# HANDBOOK OF Preparative inorganic chemistry

## VOLUME 1 · SECOND EDITION

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## From the Preface to the First Edition

For many years, the inorganic section of the "Handbook of Preparative Chemistry" by L. Vanino was a laboratory standard. By 1940, however, the third (and last) edition of the handbook was no longer in print. Rather than simply reissue the Vanino manual, the Ferdinand Enke Press projected a completely new book: in contrast to the old, the new work would be written by a number of inorganic chemists, each a specialist in the given field.

As editor, the publishers were able to obtain the services of Prof. Robert Schwartz. It was Prof. Schwartz who laid down what was to be the fundamental guideline for all subsequent work: that only those procedures were to be included which had been tested and confirmed in laboratory practice. Concerning the choice of substances, while not pretending to be exhaustive, the book would cover most of the compounds of inherent scientific interest or of importance for purposes of instruction. At the same time, it was clearly apparent that the common commercial chemicals, as well as those whose preparations require only the simplest chemical operations, need not be included.

The organization of the work took account of the broad scope and varied nature of contemporary preparative inorganic chemistry. The increasingly rigorous purity requirements, the use of unstable substances and those sensitive to air and moisture, the employment of ultralow and ultrahigh temperatures and pressures, etc., have increasingly complicated the experimental apparatus and techniques. Thus, in the introductory part (Preparative Methods) the authors have endeavored to assemble a number of experimental techniques and special apparatus that can be extended to applications much more general than the original purposes for which they were designed. This is complemented by an Index of Techniques at the end of the work. This index links the contents of Part I with the various experimental procedures distributed throughout the work. Space considerations have forced abridgments in several places. Thus, a literature reference must often take the place of a more detailed description. Occasionally, different researchers have solved a given problem by different experimental techniques. Here again a reference to the literature is in order. Naturally, the choice of preferred method is always a subjective decision of the individual experimenter. Thus, our own selection may not always seem correct or adequate to every inorganic chemist. As is customary, please forward any pertinent criticism to either the editor or publisher. It will be gratefully received.

What has been said above also holds true for Part II (Elements and Compounds) and even more so for Part III (Special Groups of Substances). In every case the decision as to inclusion or omission was dictated by considerations of available space. Here, again, the editor would be grateful for any suggestions or criticisms.

## Preface to the Second Edition

The first edition of the Handbook of Preparative Inorganic Chemistry was intended to fill a gap in the existing literature. Because it accomplished its mission so well, it has won wide respect and readership. Thus, the authors have been persuaded to issue a second, revised and enlarged edition, even though a relatively brief period has elapsed since the appearance of the first.

The present edition is much more than a revision of the previous work.

Several sections had to be completely rewritten; in a number of cases, the choice of compounds to be included has been changed; above all, recently developed processes, methods and apparatus could not be neglected. The reader will note also that several new authors have cooperated in this venture.

Thus, we are presenting what is in many respects a completely new work. Most of the preparative methods presented here have either been verified by repetition in the author's own laboratory or checked and rechecked in those of our collaborators. We trust that the reader will benefit from the improved reliability and reproducibility that this affords.

The editorial work could not have been completed without the invaluable help of Dr. H. Bärninghausen, Miss G. Boos, and my wife, Doris Brauer. Credit for the careful layout of the more than eighty new or revised drawings found in the book goes to Mrs. U. Sporkert. To all of my co-workers, advisers, colleagues and friends who have given their assistance, I wish to extend my heartfelt thanks.

## Translation Editor's Preface

The Handbook of Preparative Inorganic Chemistry by G. Brauer has been a valuable addition to the detailed preparative literature for some years largely because of the number and diversity of methods which are contained in its pages. The translation of this work, therefore, will simplify the task of synthesis for chemists whose German is less than proficient.

Because laboratory practice, as outlined in Part I of the Handbook, is in some ways different from laboratory practice in the United States a number of additions and omissions have been made in the translated text. These include: (1) the removal of the names of German suppliers and trade names and the substitution of American trade names and suppliers, the latter only occasionally, (2) conversion of German glass and ground-glass joint sizes to their American equivalents, (3) substitution throughout the text of "liquid nitrogen" for "liquid air", (4) improvement in the nomenclature where it was judged unclear. In addition, certain brief sections have been omitted or rewritten when the practice or equipment described was outmoded or so different as to be inapplicable in the United States.

It is hoped that these changes have been consistent and wise despite the diffusion of responsibility for the production of a book of this size.

Reed F. Riley

Brooklyn, New York August, 1963

## Conversion of Concentration Units

 $D_{st}$  = density of solvent

 $D_{sn}$  = density of solution

 $D_{se}^{sn}$  = density of solute

 $M_{st}$  = molecular weight of solvent

M<sub>se</sub> = molecular weight of solute

	Unit	a	Ъ	с	d
a	g./100 ml. solvent	a	b•D <sub>st</sub>	$\frac{100 \cdot c \cdot D_{st}}{(100 \cdot D_{sn}) - c}$	$\frac{100 \cdot d \cdot D_{st}}{100 - d}$
ь	g./100 g. solvent	a D <sub>st</sub>	Ъ	$\frac{100 \cdot c}{(100 \cdot D_{sn})-c}$	<u>100 • d</u> 100d
c	g./100 ml. solution	$\frac{100 \cdot a \cdot D_{sn}}{(100 \cdot D_{st}) + a}$	$\frac{100 \cdot b \cdot D_{sn}}{100 + b}$	с	d • D <sub>sn</sub>
d	g./100 g. solution (wt. %)	$\frac{100 \cdot a}{(100 \cdot D_{st}) + a}$	<u>100 • b</u> 100 + b	$\frac{c}{D_{sn}}$	d

		d	е	f
d	g./100 g. solution (wt. %)	d	e • D <sub>se</sub> D <sub>sn</sub>	$\frac{100}{1 + \left(\frac{100 - f}{f}\right)_{M_{se}}^{M_{st}}}$
e	ml./100 ml. solution (vol. %)	$\frac{d \cdot D_{sn}}{D_{se}}$	e	$\frac{100 \cdot D_{sn}/D_{se}}{1 + \left(\frac{100 - f}{f}\right) \frac{M_{st}}{M_{se}}}$
f	moles/100 moles solution (mole %)	$\frac{100}{1 + \left(\frac{100 - d}{d}\right) \frac{M_{se}}{M_{st}}}$	$\frac{100}{1 + \left(\frac{100 \cdot D_{sn}}{e \cdot D_{se}} - 1\right) \frac{M_{se}}{M_{st}}}$	f

mole fraction = moles of solute/total moles =  $\frac{f}{100}$ molality = moles of solute/1000 g. of solvent =  $\frac{10 \cdot b}{M_{se}}$ molarity = moles of solute/1000 ml. of solution =  $\frac{10 \cdot c}{M_{se}}$ 

Example: The concentration of a solution of sulfur in carbon disulfide  $(15^{\circ}C, \text{ given } D_{sn} = 1.35, D_{st} = 1.26, D_{se} = 2.07)$  is 24.0 g. S/100 ml. CS<sub>2</sub> or 19.05 g. S/100 g. CS<sub>2</sub> or 21.6 g. S/100 ml. solution or 16.0 g. S/100 g. solution or 16.0 wt. % or 10.4 vol. % or 31.2 mole %.

### Contents

FROM THE PREFACE TO THE FIRST EDITION	v
PREFACE TO THE SECOND EDITION	vi
TRANSLATION EDITOR'S PREFACE	vii
CONVERSION OF CONCENTRATION UNITS	viii

## Part | Preparative Methods

PREPARATIVE METHODS	- 3
Assembly of Apparatus	4
Glass	5
Ceramic Materials	12
Metals	17
Plastics	25
Pure Solvents	25
Mercury	27
Sealing Materials and Lubricants	28
High Temperatures	32
Low Temperatures	42
Constant Temperature	45
Temperature Measurement	49
High Vacuum and Exclusion of Air	53
Special Vacuum Systems	66
Gases	77
Liquefied Gases as Solvent Media	86
Electrical Discharges	90
Purification of Substances	91
Analysis of Purity	100
Powder Reactions	103

#### Part II

## **Elements and Compounds**

SECTION I. HYDROGEN, DEUTERIUM, WATER	111
Hydrogen H <sub>2</sub>	111
Pure Water	117
Deuterium and Deuterium Compounds	119
Deuterium D <sub>2</sub>	121

Hydrogen Deuteride HD	126
Deuterium Fluoride DF	127
Deuterium Chloride DCl	129
Deuterium Bromide DBr	131
Deuterium Iodide DI	133
Deuterium Sulfide D.S.	134
Deuterosulfuric Acid D-SO	135
Deuteroammonia ND.	137
Deuterophosphoric Acid $D_3PO_4$	138
SECTION 2. HYDROGEN PEROXIDE	140
Hydrogen Peroxide H <sub>2</sub> O <sub>2</sub>	140
SECTION 3. FLUORINE, HYDROGEN FLUORIDE	143
Fluorine F <sub>2</sub>	143
Hydrogen Fluoride HF	145
SECTION 4. FLUORINE COMPOUNDS	150
General Remarks	150
Chlorine Monofluoride ClF	153
Chlorine Trifluoride ClF <sub>2</sub>	155
Bromine Trifluoride BrF3	156
Bromine Pentafluoride BrF <sub>5</sub>	158
Iodine Pentafluoride IF 5	159
Iodine Heptafluoride IF <sub>7</sub>	160
Dioxygen Difluoride O <sub>2</sub> F <sub>2</sub>	162
Oxygen Difluoride OF	163
Chlorine Dioxide Fluoride ClO <sub>2</sub> F	165
Chlorine Trioxide Fluoride ClO <sub>2</sub> F	166
Chlorine Tetroxide Fluoride ClO, F	167
Sulfur Tetrafluoride SF	168
Sulfur Hexafluoride SF	169
Thionyl Fluoride SOF	170
Thionyl Tetrafluoride SOF	171
Sulfuryl Fluoride SO <sub>2</sub> F <sub>2</sub>	173
Trisulfuryl Fluoride S.O.F.	174
Thionyl Chloride Fluoride SOCIF	174
Sulfuryl Chloride Fluoride SO-CIF	175
Sulfuryl Bromide Fluoride SO_BrF	176
Fluorosulfonic Acid HSO.F	177
Potassium Fluorosulfinate $KSO_F$	178
Selenium Hexafluoride SeF.	179
Selenium Tetrafluoride SeF	180
Tellurium Hexafluoride TeF-	180
Nitrogen Trifluoride NF.	181
Ammonium Fluoride NH F	183
Ammonium Hydrogen Fluoride NH F . HF	183
	~~~

Nitrosyl Fluoride NOF	184
Nitrososulfuryl Fluoride FSO <sub>2</sub> NO	186
Nitryl Fluoride NO <sub>2</sub> F	186
Fluorine Nitrate NO <sub>3</sub> F	187
Phosphorus (III) Fluoride PF <sub>3</sub>	189
Phosphorus (V) Fluoride PF	190
Phosphorus Dichloride Fluoride PCl <sub>2</sub> F	191
Phosphorus Dichloride Trifluoride PCl <sub>2</sub> F <sub>2</sub>	192
Phosphorus Oxide Trifluoride POF	193
Tetrachlorophosphonium Hexafluorophosphate (V)	
$PCl_{4} \cdot PF_{6}$	193
Phosphonitrilic Fluorides (PNF <sub>2</sub> ), (PNF <sub>2</sub> ),	194
Ammonium Hexafluorophosphate (V) NH <sub>4</sub> PF <sub>6</sub>	195
Ammonium Difluorophosphate (V) NH <sub>4</sub> PO <sub>2</sub> F <sub>2</sub>	196
Potassium Hexafluorophosphate (V) KPF	196
Arsenic (III) Fluoride AsF	197
Arsenic (V) Fluoride AsF	198
Antimony (III) Fluoride SbF.	199
Antimony (V) Fluoride SbF	200
Antimony Dichloride Trifluoride SbCl. F.	200
Bismuth (III) Fluoride BiF	201
Bismuth (V) Fluoride BiF	202
Carbon Tetrafluoride CF	203
Trifluoromethane CHF.	204
Trifluoroiodomethane CIF,	205
Carbonyl Fluoride COF	206
Carbonyl Chlorofluoride COClF	208
Carbonyl Bromofluoride COBrF	210
Carbonyl Iodofluoride COIF	211
Silicon Tetrafluoride SiF	212
Trifluorosilane SiHF	214
Hexafluorosilicic Acid H-SiF	214
Germanium Tetrafluoride GeF	215
Potassium Hexafluorogermanate K-GeF	216
Tin (II) Fluoride SnF	217
Tin (IV) Fluoride SnF	217
Lead (II) Fluoride PbF	218
Lead (IV) Fluoride PbF	219
Boron Trifluoride BF	219
Fluoroboric Acid HBF	221
Sodium Fluoroborate NaBF	222
Potassium Fluoroborate KBF	223
Potassium Hydroxyfluoroborate KBF_OH	223
Nitrosvi Fluoroborate NOBF	224
Aluminum Fluoride	225
Ammonium Hexafluoroaluminate (NH .) AlF	226
Ammonium Tetrafluoroaluminate NH AIF	227

Gallium (III) Fluoride GaF <sub>3</sub>	227
Ammonium Hexafluorogallate $(NH_4)_3(GaF_6)$	228
Indium (III) Fluoride InF <sub>a</sub>	228
Ammonium Hexafluoroindate $(NH_4)_3 (InF_5) \dots \dots \dots$	229
Thallium (I) Fluoride TIF	230
Thallium (III) Fluoride T1F <sub>2</sub>	230
Beryllium Fluoride BeF,	231
Ammonium Tetrafluorobervllate $(NH_4)_2$ BeF <sub>4</sub> ,	232
Magnesium Fluoride MgF <sub>2</sub>	232
Calcium Fluoride CaF	233
Strontium Fluoride SrF.	234
Barium Fluoride BaF,	234
Lithium Fluoride LiF	235
Sodium Fluoride NaF	235
Potassium Fluoride KF	236
Potassium Hydrogen Fluoride KF • HF	237
Potassium Tetrafluorobromate (III) KBrF	237
Potassium Hexafluoroiodate (V) KIF	238
Copper (II) Fluoride CuF	238
Silver Subfluoride Ag. F	239
Silver Fluoride AgF	240
Silver (II) Fluoride $A\sigma F_{a}$	241
Zinc Fluoride ZnF	242
Cadmium Fluoride CdF	243
Mercury (I) Fluoride Hg. F.	243
Mercury (II) Fluoride HoF.	244
Scandium Fluoride ScF	245
Vttrium Fluoride VF	246
Lanthanum Fluoride LaF	246
Cerium (III) Fluoride CeF	247
Cerium (IV) Fluoride CeF.	247
Furonium (II) Fluoride FuF	248
Titanium (III) Fluoride TiF	248
Titanium (IV) Eluoride TiF $_{3}$	250
Zirconjum (IV) Fluoride $Z$ rF	251
Vanadium (III) Fluoride VE	252
Vanadium (III) Fluoride VF $_3$	252
Vanadium (V) Fluoride VF	253
Valiaululi (V) Fluoride NbE	254
Niobium (v) Fluoride Nor $\beta$ NhE	201
Tentelum (V) Elucrido TeE	200
$Tantatum (v) Fluoride Targ \dots \dots$	200
Chromitian (II) Elizarida CrE	200
Chromium (II) Fluoride CrF <sub>2</sub>	200
Chromium (III) Fluoride CrF <sub>3</sub>	201
Chromid Elucatido Cro E	200
Malabdanum (M) Elucride McE	200
worypaenum (v) Fluoride worg	209

	Tungsten (VI) Fluoride WF	260
	Uranium (IV) Fluoride UF	261
	Uranium (VI) Fluoride UF <sub>2</sub>	262
	Manganese (II) Fluoride MnF	262
	Manganese (III) Fluoride MnF	263
	Potassium Hexafluoromanganate (IV) K. MnF.	264
	Rhenium (VI) Fluoride ReF.	264
	Iron (II) Fluoride FeF	266
	Iron (III) Fluoride FeF	266
	Cobalt (II) Fluoride CoF	267
	Cobalt (III) Fluoride CoF	269
	Nickel (II) Fluoride Ni $E$	200
	Dotaggium Hovefluoronickolato (IV) K NiE	209
	Inidium (VI) Eluonido InE	209
	$\operatorname{Iridium}(v) \operatorname{Fluoride} \operatorname{IrF}_{6} \dots \dots$	270
~		07.0
SEC	TION 5. CHLORINE, BROMINE, IODINE	272
		272
	Chlorine Hydrate $Cl_2 \cdot 6 H_2 O \dots O h_2 O \dots O h_2 O \dots O h_2 O h_$	274
	Bromine Br	275
	Bromine Hydrate $Br_2 \cdot 8 H_2O$	276
	lodine I <sub>2</sub>	277
	Hydrogen Chloride HCl	280
	Hydrogen Bromide HBr	282
	Hydrogen Iodide HI	286
	Ammonium Iodide NH الم	289
	Potassium Iodide KI	290
	Iodine Monochloride IC1	290
	Iodine Monobromide IBr	291
	Iodine Trichloride ICl <sub>3</sub>	292
	Polyhalides	293
	Potassium Triiodide KI <sub>2</sub> · H <sub>2</sub> O	294
	Cesium Dichlorobromide CsBrCl,	294
	Potassium Dichloroiodide KICl	295
	Cesium Dichloroiodide CsICl	296
	Potassium Dibromoiodide KIBr	296
	Cesium Dibromoiodide CsIBr	297
	Potassium Tetrachloroiodide KICl	298
	Tetrachloroiodic Acid HICl . 4 H <sub>2</sub> O	299
	Dichlorine Oxide Cl.O	299
	Chlorine Dioxide ClO	301
	Dichlorine Hexoxide Cl-O.	303
	Dichlorine Hentoxide Cl_O	304
	Bromine Oxides	306
	Diodine Dentoxide $I_{-}$	307
	Hypochloroug Acid HClO	305
	Sodium Himachlanite NaCla, 5 H a	300
	Sodium Humohromita NaBro 5 4 0	210
	Southin hypostomic Nabro $\cdot $ $\partial$ $\Pi_2 \cup \ldots \dots \cup \ldots \cup$	910

	Dotassium Hymobromite KBrO • 3 H.O	211
	Sodium Chlorite NaClo · 3 H_O	312
	Chloric Acid HClO.	312
	Ammonium Chlorate NH ClO.	313
	Barium Chlorate Ba $(ClO_{2})_{\bullet}$ · H <sub>2</sub> O	314
	Bromic Acid HBrO	315
	Barium Bromate Ba(BrO <sub>1</sub> ) $\rightarrow$ H <sub>2</sub> O	316
	India Acid HIO.	316
	Perchloric Acid HClO.	318
	Alkaline Earth Perchlorates	320
	Nitrosvi Perchlorate NOClO	320
	Nitryl Perchlorate NO_ClO	321
	Periodic Acid H_IO_	322
	Sodium Periodates Na H-IO. NaIO.	323
	Potassium Periodate KIO.	325
	Barium Periodate Ba $H_{(IO_{1})}$	326
	Chlorine Nitrate CINO.	326
	Dinvriding inding (I) Perchlorate $[I(C_H, N)_3]Clo$	327
	Bromine (III) Nitrate $Br(NO)$	328
	Indine (III) Nitrate $I(NO)$	329
	Indine (III) Sulfate I (SO $_{3}$ ) $_{3}$	320
	Indine (III) Derchlorate $I(C(0))$	330
	Indine (III) Indate $I(IO)$ or $I_{1O}$	331
	Ovoidone (III) Sulfate (IO) SO $\cdot$ H O	333
	Dijodine Tetrovide IO $\cdot$ IO or I-O.	333
		000
SEC	TION 6. OXYGEN, OZONE	334
	Oxygen O.	334
	Ozone O	337
SEC	TION 7. SULFUR, SELENIUM, TELLURIUM	341
	Sulfur S	341
	Hydrogen Sulfide H <sub>2</sub> S	344
	Crude Sulfane H.S.	346
	Pure Sulfanes	349
	Ammonium Hydrogen Sulfide NH <sub>4</sub> HS	357
	Sodium Hydrogen Sulfide NaHS	357
	Sodium Sulfide Na <sub>2</sub> S	358
	Potassium Sulfide K <sub>2</sub> S	360
	Sodium Disulfide Na S	361
	Potassium Disulfide K-S-	363
	Potassium Trisulfide K.S.	364
	Sodium Tetrasulfide Na S	365
	Potassium Tetrasulfide K.S.	366
	Sodium Pentasulfide Na-S-	367
	Potassium Pentasulfide K.S.	367
	Potassium Hexasulfide K.S.	368

Ammonium Pentasulfide $(NH_4)_2S_5$	369
Dichloromonosulfane SCl <sub>2</sub>	370
Dichlorodisulfane S <sub>2</sub> Cl <sub>2</sub>	<b>371</b>
Dichlorotri-, -tetra-, -penta-, -hexa-, -hepta- and -octa-	
sulfane S <sub>a</sub> Cl <sub>a</sub> , S <sub>4</sub> Cl <sub>a</sub> , S <sub>5</sub> Cl <sub>a</sub> , S <sub>6</sub> Cl <sub>a</sub> , S <sub>7</sub> Cl <sub>a</sub> , S <sub>6</sub> Cl <sub>a</sub>	372
Dichlorotrisulfane S.Cl.	373
Dichlorotetrasulfane S <sub>4</sub> Cl <sub>2</sub>	375
Sulfur Tetrachloride SCl	376
Dibromodisulfane S <sub>2</sub> Br.	377
Dibromotri-,-tetra-, -penta-, -hexa-, -hepta- and -octa-	
sulfane S.Br., S.Br., S.Br., S.Br., S.Br., S.Br.	379
Lower Sulfur Oxides S.O. SO	379
Disulfur Trioxide S <sub>2</sub> O <sub>2</sub>	380
Polysulfur Peroxide (SO <sub>2</sub> )	382
Thionyl Chloride SOCI	382
Sulfuryl Chloride SO-Cl	383
Chlorosulfonic Acid HSO_Cl	385
Pyrosulfuryl Chloride S-O-Cl.	386
Thionyl Bromide SOBr-	387
Deroxymonogulfuric Acid H-SO	388
Derovydigulfurio Acid $H \subseteq O$	380
$\Delta mmonium Derovudigulfate (NH) \\ S \\ O$	300
Dotaggium Denovudigulfate K $S \cap$	392
Coholt Sulfovulate Cos $\Omega_{2}$ 3 H $\Omega_{2}$	303
Sodium Dithionito No. S. O. + 2 H. O.	303
Zina Dithionita ZnS $O$	204
Sodium Dithionato No $\mathcal{C}$ $\mathcal{O}$ $\mathcal{O}$ $\mathcal{O}$	205
Soulum Dithionate $\operatorname{Na}_{2}\operatorname{S}_{2}\operatorname{O}_{6} \circ 2\operatorname{H}_{2}\operatorname{O}_{6} \circ \ldots \circ $	207
Darrum Dimonate $BaS_2O_6 \cdot 2 \Pi_2O \dots \dots \dots \dots$	201
Potassium Tritnionate $K_{g}S_{g}U_{6}$	390
Potassium Tetratnionate $K_{3}S_{4}O_{6}$	399
Potassium Pentatnionate $K_{g}S_{g}O_{g} \cdot 1.5 \Pi_{g}O_{f} \cdot \dots$	401
Potassium Hexatnionate $K_2 S_6 O_6 \dots \dots \dots \dots \dots$	403
	405
Polythionic Acids $H_2S_XO_3$ , $H_2S_XO_6$	405
Nitrosyl Hydrogen Sulfate (NO)HSO <sub>4</sub>	406
Tetrasulfur Tetranitride $S_4N_4$	406
Tetrasulfur Dinitride $S_4N_2$	408
Disulfur Dinitride S <sub>2</sub> N <sub>2</sub>	409
Sulfur Nitride Tetrahydride $S_4(NH)_4$	411
Heptasulfur Imide S <sub>7</sub> NH.	411
$\alpha$ -Sulfanuric Chloride [OS(N)Cl] <sub>3</sub>	412
Trisulfur Dinitrogen Dioxide S <sub>3</sub> N <sub>2</sub> O <sub>2</sub>	413
Trisulfur Dinitrogen Pentoxide S <sub>3</sub> N <sub>2</sub> O <sub>5</sub>	414
Selenium Se	415
Hydrogen Selenide H <sub>2</sub> Se	418
Sodium Hydrogen Selenide NaHSe	419
Sodium Selenide, Potassium Selenide Na <sub>2</sub> Se, K <sub>2</sub> Se	421

Sodium Diselenide Na <sub>2</sub> Se <sub>2</sub>	421
Diselenium Dichloride Se <sub>2</sub> Cl <sub>2</sub>	422
Selenium Tetrachloride SeCl	423
Hexachloroselenium Salts	425
Diselenium Dibromide Se <sub>2</sub> Br <sub>2</sub>	426
Selenium Tetrabromide SeBr 4	427
Selenium Dioxide SeO	428
Selenium Oxychloride SeOCl	429
Selenous Acid (anhydrous) H <sub>2</sub> SeO <sub>3</sub>	430
Sodium Selenite Na $_{2}$ SeO $_{2}$ · 5 H $_{2}$ O	431
Selenic Acid H <sub>2</sub> SeO <sub>4</sub>	432
Sodium Selenate Na SeO	433
Sodium Selenopentathionate Na $_{2}SeS_{4}O_{4} \cdot 3H_{2}O_{5}$	434
Selenium Sulfur Trioxide SeSO	435
Selenium Nitride Se N 4	435
Tellurium Te.	437
Colloidal Tellurium Solution.	438
Hydrogen Telluride HaTe	438
Sodium Telluride. PotassiumTelluride	
Na <sub>2</sub> Te. K <sub>2</sub> Te	441
Sodium Ditelluride Na <sub>2</sub> Te <sub>2</sub>	442
Tellurium Tetrachloride TeCl 4	442
Hexachlorotellurium Salts	444
Tellurium Tetrabromide TeBr <sub>4</sub>	445
Tellurium Tetraiodide Tel	447
Tellurium Dioxide TeO2	447
Tellurous Acid HaTeO	449
Sodium Tellurite NasTeOs	449
Tellurium Trioxide TeO <sub>2</sub>	450
Telluric Acid HaTeOa.	451
Sodium Tetrahydrogentellurate (VI) Na <sub>2</sub> H <sub>4</sub> TeO <sub>4</sub>	453
Sodium Orthotellurate Na TeO	453
Sodium Telluropentathionate $Na_2 TeS_4 O_4 \cdot 2 H_2 O_5 \cdots$	454
Tellurium Sulfur Trioxide TeSO,	455
6	
SECTION 8. NITROGEN	457
Nitrogen N <sub>2</sub>	457
Ammonia NH <sub>3</sub>	460
Lithium Amide LiNH <sub>2</sub>	463
Lithium Imide Li <sub>2</sub> NH	464
Sodium Amide NaNH <sub>2</sub>	465
Hydrazinium Sulfate N <sub>2</sub> H <sub>8</sub> SO <sub>4</sub>	468
Hydrazine Hydrate $N_2H_4 \cdot H_2O_{\dots}$	469
Hydrazine N <sub>2</sub> H <sub>4</sub>	469
Hydrazoic Acid HN <sub>3</sub>	472
Azides	474
Chlorine Azide ClN3	476

Monochloramine ClNH <sub>2</sub>	. 477
Nitrogen Trichloride NCl <sub>3</sub>	. 479
Nitrogen Triiodide and Tribromide, Monobromamine	. 480
Thionyl Imide SONH	. 480
Sulfamide SO <sub>2</sub> (NH <sub>2</sub> ) <sub>2</sub>	. 482
Trisulfimide and Its Silver Salt	
$(SO_2NH)_2$ $(SO_2N)_2Ag_2 \cdot 3H_2O$	. 483
Nitrous Oxide N <sub>2</sub> O	. 484
Nitric Oxide NO.	. 485
Nitrogen Trioxide N <sub>2</sub> O <sub>3</sub>	. 487
Nitrogen Dioxide $NO_2$ , $N_2O_4$	. 488
Nitrogen Pentoxide N <sub>2</sub> O <sub>5</sub>	. 489
Nitric Acid HNO <sub>3</sub>	. 491
Hyponitrous Acid $H_2N_2O_2$	. 492
Silver Hyponitrite Ag <sub>2</sub> N <sub>2</sub> O <sub>2</sub>	. 493
Sodium Hyponitrite $Na_2N_2O_2 \cdot 9H_2O$	. 495
Nitramide NH <sub>2</sub> NO <sub>2</sub>	. 496
Hydroxylammonium Chloride (NH <sub>2</sub> OH)Cl	. 498
Hydroxylammonium Salts.	500
Hvdroxylamine NH <sub>2</sub> OH	. 501
Potassium Hydroxylamine Disulfonate	
HON(SO <sub>2</sub> K) 2	503
Potassium Dinitrososulfite $K_2SO_3 \cdot (NO)_2$	504
Potassium Nitrosodisulfonate ON(SO <sub>3</sub> K) <sub>2</sub>	. 504
Potassium Nitrilosulfonate $N(SO_3K)_3 \cdot 2H_2O_5$	506
Potassium Imidosulfonate HN(SO <sub>3</sub> K) <sub>2</sub>	506
Potassium Amidosulfonate HaNSO3K	507
Potassium Chloroimidosulfonate ClN(SO <sub>3</sub> K) <sub>2</sub> ,	508
Amidosulfonic acid H <sub>2</sub> NSO <sub>2</sub> H	508
Potassium Hydrazinedisulfonate H <sub>2</sub> N <sub>2</sub> (SO <sub>2</sub> K) <sub>2</sub>	. 509
Potassium Azodisulfonate N <sub>2</sub> (SO <sub>2</sub> K) <sub>2</sub>	510
Hydroxylamineisomonosulfonic Acid NH <sub>2</sub> SO <sub>4</sub>	510
Nitrosyl Chloride NOC1	511
Nitrosvl Bromide NOBr.	513
Nitryl Chloride NO <sub>2</sub> Cl.	. 513
Sodium Nitrosvl NaNO.	514
Sodium Nitroxylate Na <sub>2</sub> NO <sub>2</sub>	515
Sodium Hyponitrate Na <sub>2</sub> N <sub>2</sub> O <sub>3</sub>	517
SECTION 9. PHOSPHORUS	518
White Phosphorus P <sub>4</sub>	518
Red Phosphorus	. 519
Black Phosphorus.	522
Colloidal Phosphorus	524
Phosphine and Diphosphine PH <sub>3</sub> P <sub>2</sub> H <sub>4</sub>	525
Sodium Dihydrogenphosphide NaPH2	530
Phosphonium Iodide PH <sub>4</sub> I	531

	Thiophosphoryl Chloride PSCl <sub>3</sub>	532
	Phosphoryl (V) Bromide POBra	534
	Thiophosphoryl (V) Bromide PSBr	535
	Diphosphoric Acid Tetrachloride $P_2O_3Cl_4$	536
	Diphosphorus Tetraiodide P <sub>2</sub> I <sub>4</sub>	539
	Phosphorus (III) Iodide PI	540
	Phosphorus (V) Oxide $P_2O_4$ ( $P_4O_{10}$ )	541
	Orthophosphoric Acid H <sub>2</sub> PO <sub>4</sub>	543
	Sodium Dihydrogen Phosphate $NaH_2PO_4 \cdot 2H_2O_5 \cdots O_4$	544
	Potassium Phosphate $K_2PO_4 \cdot 8H_2O$	545
	Hydroxyapatite $Ca_{10}(PO_4)_{e}(OH)_{2}$ .	545
	Condensed Orthophosphates	546
	Polyphosphates	549
	Metaphosphates	552
	Orthophosphorous Acid H <sub>3</sub> PO <sub>3</sub>	554
	Hypophosphorous Acid H <sub>3</sub> PO <sub>2</sub>	555
	Barium Hypophosphite $Ba(H_2PO_2)_2 \cdot H_2O_1$	557
	Hypophosphoric Acid $H_4P_2O_4$	558
	Disodium Dihydrogen Hypophosphate	
	$Na_2H_2P_2O_6 \cdot 6H_2O$	560
	Tetrasodium Hypophosphate Na ${}_{4}P_{2}O_{6} \cdot 10 H_{2}O \dots$	561
	Barium Dihydrogen Hypophosphite $BaH_2P_2O_6 \cdot 2H_2O_{\cdot}$	562
	Potassium Peroxydiphosphate K <sub>4</sub> P <sub>2</sub> O <sub>8</sub>	562
	Phosphorus Trisulfide $P_4S_3$	563
	Phosphorus Pentasulfide P <sub>4</sub> S <sub>5</sub>	565
	Phosphorus Heptasulfide $P_4S_7$	566
	Diphosphorus Pentasulfide P <sub>2</sub> S <sub>5</sub>	567
	Monothiophosphoric Acid $H_3PO_3S$	568
	Sodium Monothiophosphate Na <sub>3</sub> PO <sub>3</sub> S · 12 H <sub>2</sub> O	569
	Sodium Dithiophosphate $Na_3PO_2S_2 \cdot 11H_2O$	570
	Barium Dithiophosphate $Ba_3(PO_2S_2)_2 \cdot 8 H_2O \dots$	571
	Sodium Trithiophosphate $Na_3POS_3 \cdot 11 H_2O \dots$	571
	Sodium Tetrathiophosphate Na <sub>3</sub> PS <sub>4</sub> .8 H <sub>2</sub> O	572
	Tetraphosphorus Triselenide P <sub>4</sub> Se <sub>3</sub>	573
	Triphosphorus Pentanitride P <sub>3</sub> N <sub>5</sub>	574
	Phosphonitrilic Chlorides (PNCl <sub>2</sub> ) <sub>n</sub>	575
	Phosphonitrilic Bromides (PNBr <sub>2</sub> ) <sub>n</sub>	578
	Monoamidophosphoric Acid H <sub>2</sub> PO <sub>3</sub> NH <sub>2</sub>	579
	Disodium Monoamidophosphate $Na_2PO_3NH_2 \cdot 6H_2O$	581
	Diamidophosphoric Acid HPO <sub>2</sub> (NH <sub>2</sub> ) <sub>2</sub>	582
	Phosphoryl Triamide PO(NH <sub>2</sub> ) <sub>3</sub>	584
	Thiophosphoryl Triamide PS(NH <sub>2</sub> ) <sub>3</sub>	587
	Pyrophosphoryl Tetramide $P_2O_3(NH_2)_4$	588
	Tetrasodium Imidodiphosphate Na ${}_{4}P_{2}O_{6}NH \cdot 10 H_{2}O \dots$	589
SECT	ION IO. ARSENIC, ANTIMONY, BISMUTH	591
	Arsenic As	591

Arsine AsH <sub>3</sub>	593
Sodium Dihydrogen Arsenide NaAsH <sub>2</sub>	595
Arsenic Trichloride AsCl <sub>3</sub>	596
Arsenic Tribromide AsBr <sub>3</sub>	597
Arsenic Triiodide AsI3	597
Arsenic Diiodide AsI2	598
Diarsenic Trioxide $As_2O_3(As_4O_4)$	600
Orthoarsenic Acid H <sub>2</sub> AsO <sub>4</sub>	601
Sodium Dihydrogen Orthoarsenate NaH <sub>2</sub> AsO <sub>4</sub> •H <sub>2</sub> O	602
Ammonium Orthoarsenate (NH 4) 3AsO 4 · 3 H O	602
Tetraarsenic Tetrasulfide As <sub>4</sub> S <sub>4</sub>	603
Diarsenic Pentasulfide As <sub>2</sub> S <sub>5</sub>	603
Ammonium Thioarsenate (NH <sub>4</sub> ) <sub>3</sub> AsS <sub>4</sub>	604
Sodium Thioarsenate Na AsS 4.8 H2O	604
Sodium Monothioorthoarsenate Na AsO S · 12 H 20	605
Sodium Dithioorthoarsenate Na AsO S. · 11 H.O.	605
Antimony Sb	606
Stibine (Antimony Hydride) SbH	606
Antimony (III) Chloride SbCl	608
Antimony (V) Chloride SbCl	610
Antimony (III) Oxide Chloride SbOC1	611
Hexachloroantimonic (V) Acid HSbCl <sub>a</sub> • 4.5 H <sub>2</sub> O	611
Nitrosvl Chloroantimonate (V) NO(SbCl <sub>e</sub> )	612
Antimony (III) Bromide SbBr	613
Antimony (III) Iodide Sbla	614
Ammonium Hexabromoantimonate (IV) (NH <sub>4</sub> ) SbBr <sub>4</sub>	615
Antimony (III) Oxide Sb <sub>2</sub> O <sub>2</sub>	615
Antimony (V) Oxide Sb <sub>2</sub> O <sub>2</sub>	616
Hydrated Antimony (V) Oxide Sb <sub>2</sub> O <sub>s</sub> (H <sub>2</sub> O),	617
Diantimony Tetroxide $Sb_2O_4$	618
Antimony (ITT) Sulfate Sb <sub>2</sub> (SO <sub>4</sub> )	618
Antimony (III) Oxide Sulfate (SbO) 2SO 4	619
Sodium Thioantimonate (V) Na SbS 4 9 H 20	619
Bismuth Bi	620
Bismuth (III) Chloride BiCl	621
Bismuth Dichloride BiCl,	622
Bismuth Oxide Chloride BiOCl	622
Bismuth (III) Bromide BiBra	623
Bismuth Oxide Bromide BiOBr	624
Bismuth (III) Iodide Bil.	624
Bismuth Oxide Iodide BiOI	625
Bismuth Oxide Nitrite BiONO <sub>2</sub>	626
Bismuth (III) Phosphate BiPO	626
Bismuth (III) Borate $BiBO_2 \cdot 2H_2O_2 \cdot \dots \cdot \dots \cdot \dots \cdot \dots$	627
Sodium Bismuthate KBiO	627
Potassium Bismuthate KaBiO	628
Dibismuth Tetroxide $Bi_2O_4$	629

SECTION II. CARBON	630
A) ELEMENTAL CARBON	630
Pure Carbon	630
Special Carbon Preparations	631
Surface Compounds of Carbon	633
B) GRAPHITE COMPOUNDS	635
Alkali Graphite Compounds	635
Alkali Ammine Graphite Compounds	637
Graphite Oxide	638
Carbon Monofluoride	640
Tetracarbon Monofluoride	641
Graphite Salts	642
Bromine Graphite	643
Metal Halide Graphite Compounds	644
C) VOLATILE CARBON COMPOUNDS	645
Carbon Monoxide CO	645
Carbon Dioxide CO <sub>2</sub>	647
Tricarbon Dioxide $\tilde{C}_3 O_3$	648
Carbonyl Chloride COCl,	650
Carbon Disulfide CS,	652
Tricarbon Disulfide $\tilde{C}_3S_2$	653
Carbonyl Sulfide COS	654
Carbonyl Selenide COSe	655
Carbon Diselenide CSe <sub>2</sub>	656
Hydrogen Cyanide HCN	658
Cyanogen (CN) <sub>2</sub>	660
Cyanogen Chloride CNC1	662
Cyanogen Bromide CNBr	665
Cyanogen Iodide CNI	666
Cyanic Acid HNCO	667
Hydrogen Thiocyanate HNCS	669
Thiocyanogen (SCN) <sub>2</sub>	671
Ammonium Trithiocarbonate $(NH_4)_2CS_3$	674
Barium Trithiocarbonate BaCS <sub>3</sub>	674
SECTION 12. SILICON AND GERMANIUM	676
Silicon Si	676
Silanes SiH <sub>4</sub> (Si <sub>2</sub> H <sub>6</sub> , Si <sub>3</sub> H <sub>6</sub> )	679
Polysilanes $(SiH)_X$ , $(SiH_2)_X$ .	681
Silicon Tetrachloride SiCl <sub>4</sub>	682
Higher Silicon Chlorides	684
Silicon Tetrabromide SiBr <sub>4</sub>	686
Silicon (II) Bromide SiBr <sub>2</sub>	687
Silicon Tetraiodide Sil <sub>4</sub>	689
Chlorosilanes SiHCl <sub>3</sub> , SiH <sub>2</sub> Cl <sub>2</sub> , SiH <sub>3</sub> Cl	691
Tribromosilane SiHBr <sub>3</sub>	692
Dimethyldichlorosilane (CH <sub>3</sub> ) <sub>2</sub> SiCl <sub>2</sub>	694

Chlorosiloxanes Si (O (Cl., SinOn, Cl., Cl., Cl., Cl., Cl., Cl., Cl., Cl.					. 69	5
Silicon Monoxide SiO					. 69	6
Silicic Acids					. 69'	7
Silicon Oxyhydride HaSiaOa					. 69	9
Silicon Disulfide SiS					. 70	0
Silicon Tetraacetate Si(CH-COO)					. 70	1
Silicon Cyanate and Silicon Isocyanate	•	•••	•••	•	• • • •	-
$Si(OCN)$ $\leq Si(NCO)$					. 70	2
Tetraethoyygilane Tetramethoyygilane	•	• •	•••	•	• • • •	-
$Si(OC_{*}H_{*})$ , $Si(OCH_{*})$ ,					70	2
Silicates	•	••	•••	•	70	4
Germanium	•	••	• •	•	70	ĥ
Germanium (IV) Ovide GeO	• •	••	• •	•	70	6
Germanium (II) Oxide GeO	• •	••	• •	•	- 10. 71	1
Metallia Cormanium Co	• •	••	• •	•	• • • ± · 714	r o
Cermonium Hydrides CeH (Ce H Ce H)	• •	••	• •	•	• 114 711	e Q
Community (III) Chlorido CoCl	• •	• •	••	•	• 714 7714	5
Germanium Dichlorido GeCl $4 \dots \dots \dots$	• •	••	••	•	• 710 710	6
Commanium (IV) Promido CoPr	• •	• •	•••	•	• (±) 1710	0
Germanium $(IV)$ Indide Gebr <sub>4</sub>	• •	• •	• •	•	• (10 1710	D D
Germanium $(1v)$ foliae $GeI_4$	• •	• •	• •	•	• (1) 79/	9 0
	• •	••	• •	•	• 14 79	1
Methodagermane	• •	• •	••	•	• 72.	1
Metnylgermanium Trilodide Ch <sub>3</sub> Gel <sub>3</sub>	• •	• •	•••	•	· 722	ຊ ດ
Germanium Nitride $Ge_3N_4$	• •	• •	• •	•	. 74	4 0
Germanium Disulfide Ges <sub>2</sub>	• •	• •	••	•	. 72.	о Л
	• •	••	••	•	• 74 79	±
Tetraethoxygermane $Ge(OU_2H_5)_4$	• •	• •	• •	•	. 72;	Э С
Germanium Tetraacetate Ge(CH <sub>3</sub> COO) <sub>4</sub>	• •	• •	••	•	. 720	Ð
SECTION 12 THE AND LEAD					79	7
Tin Sn	• •	••	••	•	• 12 79'	7
Tin (II) Chlonido SnCl	• •	••	•••	•	· 12 799	2
Tin (II) Chlorido SnCl $\mathbf{x}_1$	• •	•	••	•	· 120 790	ر ۵
Howehle restance Acid H SnCl & 6 H O	• •	• •	• •	•	· 14: 721	2 0
Ammenium Heveehlevestannate. Detaggium	• •	• •	• •	•	. 10	9
Howeshlengtennets (NH) SpC1 K SpC1					79	1
$\frac{1}{2} \frac{1}{2} \frac{1}$	• •	• •	• •	•	. 10. 794	r o
Tin (II) Bromide SnBrg	• •	•	• •	•	. 734 799	≙ 0
Tin (IV) Bromide $\operatorname{SnBr}_4$	• •	• •	• •	•	. 73.	⊃ ∧
Tin (II) loalide $\operatorname{Snl}_2$	• •	•	• •	•	. 734 794	±
Tin (IV) loaide $sni_4$	• •	•	• •	•	. 730	2
Tin (II) Oxide SnO. $\ldots$	• •	•	• •	•	. 730	2
Stannic Acids $SnO_2 \cdot nH_2O_1$	• •	•	• •	•	. 737	1
Sodium Orthostannate $Na_4SnO_4$	• •	•	• •	•	. 739	1
Tin (II) Sulfide SnS	• •	•	• •	•	. 739	•
Tin (IV) Sulfide SnS <sub>2</sub>	• •	•	• •	•	. 74]	L
Sodium Metathiostannate Na <sub>2</sub> SnS <sub>3</sub> · 8 H <sub>2</sub> O	• •	•	• •	•	. 742	2
Sodium Tetrathiostannate (IV) Na <sub>4</sub> SnS <sub>4</sub> · 18 H <sub>2</sub> O	• •	•	••	•	. 743	3

Tin (IV) Sulfate $Sn(SO_4)_2 \cdot 2H_2O_1$	744
Tetramethyltin $Sn(CH_3)_4$	744
Tetraethyltin $Sn(C_2H_2)_4$	746
Tin (IV) Acetate Sn(CH <sub>3</sub> COO) <sub>4</sub>	747
Lead Pb	748
Lead (IV) Chloride PbCl <sub>4</sub>	750
Ammonium Hexachloroplumbate (NH <sub>4</sub> ), PbCl	751
Potassium Hexachloroplumbate K.PbCl	753
Potassium Iodoplumbite KPbI • 2 HoO	754
Lead (II. IV) Oxide Pb <sub>3</sub> O <sub>4</sub>	755
Lead (IV) Oxide PbO	757
Sodium Metaplumbate Na PbO	758
Sodium Orthoplumbate Na PbO	759
Calcium Orthoplumbate Ca. PbO	760
Lead Sulfide PbS	760
Lead (IV) Sulfate Pb(SO <sub>4</sub> )	761
Lead Azide Pb(Na)	763
Tetramethyllead Pb(CH <sub>2</sub> )	763
Tetraethyllead $Pb(C_H_)$	765
Neutral and Basic Lead Carbonate	
$PhCO_{2}$ , 2 $PhCO_{2}$ • $Ph(OH)_{2}$	766
Lead (IV) Acetate $Pb(CH_2COO)$	767
Lead Thiocvanate Pb(SCN).	769
SECTION 14. BORON	770
Boron	770
Aluminum Boride AlB <sub>2</sub> , AlB <sub>12</sub>	772
Diborane B.H.	773
Lithium Borohydride LiBH	775
Sodium Borohydride NaBH	776
Sodium Trimethoxyborohydride NaHB(OCH <sub>3</sub> ) <sub>3</sub>	777
Borine Trimethylaminate BH <sub>3</sub> · N(CH <sub>3</sub> ) <sub>3</sub>	778
Borazole (HBNH)	779
s-Trichloroborazole (ClBNH)	779
Boron Trichloride BCla.	780
Boron Tribromide BBra	781
Boron Triiodide BI <sub>3</sub>	782
Boron Trifluoride Dihydrate BF3 • 2 H <sub>2</sub> O	784
Dihvdroxyfluoroboric Acid H[BF.(OH)]	784
Boron Trifluoride Ammoniate BF3 • NH3	785
Boron Trifluoride Etherate $BF_2 \cdot O(C_2H_2)_2$	786
Boron (III) Oxide B <sub>2</sub> O <sub>3</sub>	787
Boron (III) Sulfide B <sub>2</sub> S <sub>3</sub>	788
Boron Nitride BN.	789
Sodium Orthoborate Na <sub>3</sub> BO <sub>3</sub>	790
Metaboric Acid HBO	791
Sodium Metaborate NaBO	791

Sodium Tetraborate $Na_2B_4O_7$	793
Sodium Pentaborate NaB <sub><math>a</math></sub> O <sub><math>a</math></sub> $\cdot$ 5 H <sub><math>a</math></sub> O	795
Sodium Perborate NaBO, • 4 H <sub>2</sub> O	795
Lindemann Glass (Lithium Beryllium Borate)	796
Boron Phosphate BPO <sub>4</sub>	796
Boron Arsenate BAsO	797
Boron Methoxide B(OCH <sub>2</sub> ) <sub>2</sub>	797
Trimethylboron B(CH <sub>3</sub> ) <sub>3</sub>	798
Triethylboron $B(C_{-}H_{-})_{2}$	799
Trimethylborazine (CH <sub>3</sub> BO) <sub>3</sub>	800
Tri-n-Butylboroxine (n-C <sub>4</sub> H <sub>2</sub> BO) <sub>3</sub>	801
n-Butylboronic Acid $n-C_4H_8B(OH)_8$	801
n-Butylboron Difluoride n-C <sub>4</sub> H <sub>6</sub> BF <sub>2</sub>	802
Sodium Tetraphenylborate $Na[B(C_eH_e)_4]$	803
SECTION 15. ALUMINUM	805
Lithium Aluminum Hydride LiAlH	805
Calcium Aluminum Hydride Ca(AlH <sub>4</sub> )	806
Polymeric Aluminum Hydride $(AlH_3)_n \cdot x O(C_2H_5)_2 \dots$	807
Aluminum Chlorohydride Al <sub>2</sub> Cl <sub>3</sub> H <sub>3</sub>	808
Aluminum Hydride Trimethylaminate	
$A1H_3 \cdot 2 N(CH_3)_3$ , $A1H_3 \cdot N(CH_3)_3$	809
Diethylaluminum Bromide $Al(C_2H_5)_2Br$	809
Triethylaluminum $Al(C_2H_5)_3$	810
Triethylaluminum Etherate $Al(C_2H_5)_3 \cdot O(C_2H_5)_2 \dots$	811
Diethylaluminum Hydride $Al(C_2H_5)_2H$	811
Aluminum Chloride AlCl <sub>3</sub>	812
Aluminum Bromide AlBr <sub>3</sub>	813
Aluminum Iodide AlI <sub>3</sub>	814
Aluminum Chloride Hydrate AlCl <sub>3</sub> · 6 H <sub>2</sub> O	815
Sodium Tetrachloroaluminate NaAlCl <sub>4</sub>	816
Tetrachloroaluminic Acid Dietherate $HAlCl_4 \cdot 2 O(C_2H_5)_2$	816
Aluminum Chloride Ammoniate AlCl <sub>3</sub> · NH <sub>3</sub> · · · · · · ·	817
Aluminum Chloride-Sulfur Dioxide Adduct AlCl <sub>3</sub> .SO <sub>2</sub>	817
Aluminum Chloride-Thionyl Chloride Adduct	
$Al_{2}Cl_{6} \cdot SOCl_{2} \dots \dots$	818
Aluminum Chloride-Phosphorus Pentachloride Adduct	
$AlCl_3 \cdot PCl_5 \dots \dots$	818
Aluminum Bromide-Hydrogen Sulfide Adduct	
$AlBr_3 \cdot H_2S$	819
Aluminum Iodide Hexaammoniate AlI <sub>3</sub> · 6 NH <sub>3</sub>	819
Aluminum Hydroxide	820
Aluminum Oxide.	822
Aluminum Sulfide Al <sub>2</sub> S <sub>3</sub>	823
Aluminum Sulfite	824
Aluminum Selenide Al <sub>2</sub> Se <sub>3</sub>	825
Aluminum Telluride Al <sub>2</sub> Te <sub>3</sub>	826

Aluminum Nitride AlN	827
Lithium Aluminum Nitride Lia AlNa	828
Aluminum Azide Al(N <sub>2</sub> ) <sub>3</sub>	829
Aluminum Phosphide AlP	829
Lithium Aluminum Phosphide Li <sub>3</sub> AlP <sub>2</sub>	830
Aluminum Orthophosphate AlPO	831
Aluminum Arsenide AlAs.	831
Aluminum Carbide Al $_{4}C_{3}$ .	832
Lithium Aluminum Cyanide LiAl(CN)	833
Aluminum Methoxide Al(OCH <sub>3</sub> ) <sub>3</sub>	833
Aluminum Ethoxide $Al(OC_2H_2)_3$	834
Aluminum Triethanolaminate $Al(OC_2H_4)_3N_{1}$	835
Aluminum Acetate Al(O <sub>2</sub> CCH <sub>3</sub> ) <sub>3</sub>	835
Aluminum Acetylacetonate $Al(C_{p}H_{7}O_{2})_{3}$	836
SECTION 16. GALLIUM, INDIUM, THALLIUM	837
Gallium Ga	837
Trimethylgallium, Tetramethyldigallane, Digallane	
$Ga(CH_3)_3$ , $Ga_2H_2(CH_3)_4$ , $Ga_2H_6$	840
Lithium Tetrahydrogallate LiGaH <sub>4</sub>	842
Gallium (III) Chloride GaCl <sub>3</sub>	843
Gallium (III) Bromide GaBr <sub>3</sub>	845
Gallium (III) Iodide GaI <sub>3</sub>	846
Gallium (II) Chloride and Gallium (II) Bromide GaCl <sub>2</sub> ,	
GaBr <sub>2</sub>	846
Gallium Hydroxide Ga(OH) 3, GaO(OH)	847
Gallium (III) Oxide $\alpha$ -Ga <sub>2</sub> O <sub>3</sub> , $\beta$ -Ga <sub>2</sub> O <sub>3</sub>	848
Gallium (I) Oxide Ga <sub>2</sub> O	849
Gallium (III) Sulfide Ga <sub>2</sub> S <sub>3</sub>	850
Gallium (II) Sulfide GaS	851
Gallium (I) Sulfide Ga <sub>2</sub> S	852
Ammonium Gallium (III) Sulfate $NH_4Ga(SO_4)_2 \cdot 12 H_2O_{}$	854
Gallium Selenide $Ga_2Se_3$ , GaSe, $Ga_2Se$	854
Gallium Telluride Ga <sub>2</sub> Te <sub>3</sub> , GaTe	855
Gallium Nitride GaN	855
Gallium Nitrate $Ga(NO_3)_3$	856
Gallium Phosphide, Arsenide and Antimonide GaP, GaAs,	
GaSb	857
	857
Indium (III) Chloride InCl <sub>3</sub>	858
Indium (III) Bromide $\ln Br_3$	859
Indium (III) Iodide Ini <sub>3</sub> indiada India India	000
Indium (II) Unioride, Bromide and Iodide InUl <sub>2</sub> , InBr <sub>2</sub> , Inl <sub>2</sub>	100
Indium (1) Unioride, Bromide and Iodide InCl, InBr, InI	002
Indium Hydroxide In(OH) 3	002
$\begin{array}{c} \text{Indiana} (\text{III}) \text{ Oxide In}_2 \text{ O}_3  \dots  \dots  \dots  \dots  \dots  \dots  \dots  \dots  \dots  $	000
	003

Indium Sulfides In <sub>2</sub> S <sub>3</sub> , InS, In <sub>2</sub> S	864
Indium Selenides and Tellurides In <sub>2</sub> Se <sub>3</sub> , InSe, In <sub>2</sub> Se and	
$In_2Te_3$ , $InTe$ , $In_2Te$	865
Indium Nitride InN	866
Indium Phosphide, Arsenide and Antimonide InP, InAs,	
InSb	867
Thallium Tl	867
Thallium (I) Chloride, Bromide and Iodide TlCl, TlBr, TlI	869
Thallium (III) Chloride TlCl <sub>2</sub> , TlCl <sub>3</sub> · 4 H <sub>2</sub> O	870
Tetrachlorothallium (III) Acid $H(T C _{4}) \cdot 3 H_{2}O_{2}$	872
Thallium (I) Tetrachlorothallate (III) Tl(TlCl <sub>4</sub> )	872
Thallium (I) Hexachlorothallate (III) Tl <sub>2</sub> (TlCl <sub>2</sub> )	873
Potassium Hexachlorothallate (III) K <sub>2</sub> (TlCl <sub>a</sub> ) · 2 H <sub>2</sub> O	873
Potassium Pentachloroaquothallate (III)	
$K_{2}(TlCl_{e}H_{2}O) \cdot H_{2}O$	874
Cesium Nonachlorodithallate (III) Cs <sub>2</sub> (Tl <sub>2</sub> Cl <sub>2</sub> )	874
Thallium (III) Bromide TlBro: 4 HoO	874
Thallium (I) Tetrabromothallate (III) Tl(TlBr <sub>4</sub> )	875
Thallium (I) Hexabromothallate (III) Tl <sub>a</sub> (TlBr <sub>a</sub> )	875
Rubidium Hexabromothallate (III) $Bb_{0}(T Br_{0}) \cdot \frac{8}{7} H_{0}O_{0}$	876
Thallium Trijodide TII • In Tilou and the second	876
Thallium (I) Oxide $Tl_{0}O$	877
Thallium (I) Hydroxide TIOH	877
Thallium (II) Oxide $Tl_2O_2$ , $Tl_2O_2 \cdot x H_2O_2$	879
Thallium Sulfides	880
Thallium (I, III) Selenide $Tl_{2}Se \cdot Tl_{2}Se_{2}$ , $TlSe_{2}$ , $TlSe_{2}$	881
Thallium (I) Sulfate Tl <sub>2</sub> SO <sub>4</sub>	881
Disulfatothallic (III) Acid HTI(SO <sub>4</sub> ) a • 4 H <sub>2</sub> O	. 882
Thallium (III) Hydroxide Sulfate Tl(OH)SO4 • 2 HoO	882
Thallium (D) Nitride TLN	883
Thallium (I) Nitrate TINO.	883
Thallium (I) Carbonate TlaCO	884
Thallium (ID Formate, Thallium (D Malonate, Clerici's	
Solution	884
SECTION 17. ALKALINE EARTH METALS	887
Bervllium Be	887
Bervllium Chloride BeCl <sub>2</sub>	889
Bervllium Bromide BeBra	891
Bervllium Iodide Bela	892
Beryllium Oxide and Beryllium Carbonate BeO. BeCO.	893
Bervllium Hydroxide Be(OH)	894
Sodium Bervilates	895
Bervllium Sulfide BeS	895
Bervllium Selenide and Bervllium Telluride BeSe. BeTe.	897
Bervllium Nitride Be No.	898
Bervllium Azide Be(N <sub>a</sub> )	899

Bervllium Carbides Be <sub>2</sub> C, BeC <sub>2</sub>	899
Bervllium Acetate Be(CH <sub>2</sub> COO) <sub>2</sub>	901
Basic Bervllium Acetate Be <sub>4</sub> O(CH <sub>2</sub> COO) <sub>6</sub>	901
Magnesium Mg	903
Magnesium Hydride MgH <sub>2</sub>	905
Magnesium Chloride MgCl <sub>2</sub>	905
Magnesium Bromide MgBro	909
Magnesium Iodide MgI	910
Magnesium Oxide MgO	911
Magnesium Hydroxide Mg(OH)	912
Magnesium Sulfide MgS.	913
Magnesium Selenide MgSe	915
Magnesium Telluride MgTe	915
Magnesium Nitride Mg.No.	916
Magnesium Azide $Mg(N_s)$	917
Magnesium Phosphide and Magnesium Arsenide Mg.P.	
Mo As	917
Magnesium Carhides MgCo MgoC	920
Magnesium Silicide Mg_Si	921
Magnesium Germanide Mg_Ge	922
Calcium Strontium Barium Metals	922
Calcium Strontium and Barium Hydrides Callo, SrHo, BaHo	929
Calcium, Strontium, Barium Halides	930
Calcium Oxide CaO	931
Strontium Oxide SrO.	932
Barium Oxide BaO	933
Calcium Hydroxide Ca(OH)	934
Strontium Hydroxide $Sr(OH)_2 \cdot 8 H_2O$ , $SrO \cdot 9 H_2O$	935
Calcium, Strontium, Barium Peroxides CaO <sub>2</sub> , SrO <sub>2</sub> , BaO <sub>2</sub>	936
Calcium, Strontium, Barium Sulfides CaS, SrS, BaS	938
Calcium, Strontium, Barium Selenides CaSe, SrSe, BaSe,	939
Calcium, Strontium, Barium Nitrides Ca.No. Sr.No. Ba.No.	940
Barium Azide Ba(Na)	942
Calcium Phosphide Ca Pa	942
Calcium Carbide CaC <sub>2</sub>	943
Calcium Cvanamide CaCN <sub>2</sub>	946
Calcium Silicides CaSi. CaSia	946
Calcium Germanide CaGe	948
SECTION 18. ALKALI METALS	950
Alkali Metal Compounds from Minerals	950
Free Alkali Metals	956
Alkali Hydrides NaH, KH, RbH, CsH and LiH	971
Alkali Metal Oxides Li <sub>2</sub> O, Na <sub>2</sub> O, K <sub>2</sub> O, Rb <sub>2</sub> O, Cs <sub>2</sub> O	974
Lithium and Sodium Peroxides $Li_2O_2$ and $Na_2O_2$	979
Alkali Dioxides	980
Lithium Hydroxide LiOH • H <sub>2</sub> O, LiOH	982

Rubidium and Cesium Hydroxides RbOH, CsOH	983 984
Phosphides, Arsenides, Antimonides and Bismuthides of	001
Alkali Metals from the Elements	985
Sodium and Lithium Carbides $Na_2C_2$ , $Li_2C_2$	987
Alkali Metal Carbonates of Highest Purity	987
Silicides and Germanides of Alkali Metals from the Ele-	
ments NaSi, KSi, RbSi, CsSi, NaGe, KGe, RbGe, CsGe.	989
FORMULA INDEX	993

## Part I Preparative Methods

### Preparative Methods

#### P. W. SCHENK AND G. BRAUER

This part of the book describes special methods and devices for inorganic preparations. We do not intend to present a comprehensive, thorough compilation of all the known methods of preparative inorganic chemistry, such as given in handbooks. An enterprise of that kind would require too much space, and the appropriate books are already available. Even through the several-volume treatise by Stock, Staehler, Tiede and Richter is by now partly outdated, many references, methods and descriptions of apparatus, useful for solving experimental problems, can be found in specialized books, such as those by Von Angerer, Dodd and Robinson, Grubitsch, Klemenc, Kohlrausch, Lux and Ostwald-Luther [1], to name but a few. These texts can thus be consulted when the need arises.

In Part I, only a more or less subjective selection of methods and devices is presented. This selection was governed by certain principles. Increased emphasis on greater purity of preparations and the advent of extreme experimental conditions have imposed more rigorous demands on the experimental equipment. Porcelain dishes and beakers must increasingly be complemented or replaced by more complicated apparatus for the preparation of unstable or oxidizable substances. Such special demands placed on individual preparatory steps have often led to the development of general procedures which can be applied to a larger number of preparations than was originally contemplated. An effort has been made to extract such standard methods and techniques from later sections and to summarize them in this first part. Whenever a too detailed description had to be omitted because of space limitations, at least the original literature reference is given. In addition to brief descriptions of the more commonly used and well-known special equipment, an attempt has also been made to describe some of the experimental "art," namely, those little tricks and short-cuts which with the passage of time have become traditional in almost every laboratory, but which somehow never seem to find their way into the literature.

#### Assembly of Apparatus

The classic Bunsen support with its clamps and brackets is still the most frequently used framework for assembling apparatus. There are various newer variations of it which eliminate the movement of the clamps when the brackets are tightened.

It is best to assemble a permanent support so that the entire structure can be easily carried about without having to dismantle it each time and so that it can be set aside when not in use. Such an arrangement is especially useful with the most commonly used pieces of apparatus, e.g., pump assemblies consisting of forepump, mercury traps and vacuum measuring instruments, or apparatus used for the preparation, purification and drying of inert or other frequently used gases. To construct more extensive assemblies, it is best to interconnect individual uprights with round steel rods 13 mm. in diameter, and to increase the stability of the whole, the uprights are fastened to similar rods, cemented into the wall. It is also very helpful to attach strong wooden strips, about 10 cm. wide, horizontally along the wall above the working benches (one strip about 30 cm., the other about 80 cm. above the bench surface). The rods holding the uprights in place can then be screwed into wall receptacles (1/4'' size, available in hardware supply stores)which are fastened to the wooden strips. These round wall receptacles can also be fastened with screws to the work bench to hold the vertical rods, thus replacing the base plate of the support. The cross braces fastened to the wall, or else suitable clamps, allow the work bench supports to be eliminated, and the entire apparatus can then be mounted directly on the wall. This has the considerable advantage of leaving the table space free, so that it can be kept clean more easily, and so that spilled mercury can be readily wiped up. If the apparatus is very tall, a "gallows" frame (Fig. 1) can be used, mounted on a table about 60 cm, above the floor. This frame is free standing and, as a result, the experimental apparatus can easily be reached from all sides. Similar structures can be built on the free-standing center benches of the laboratory by attaching four vertical rods to the two short sides of a bench and connecting them horizontally with matching round rods. Suitable perforated structural steel angles with corresponding bolts and nuts are available for the various setups, even those built up from the floor. These perforated angles can be assembled into very stable structures resembling those which children build from Erector sets. Additional suggestions and details about frame materials can be found in G. C. Mönch [2]. In assembling the apparatus, special care is required in selecting the right location and the proper apparatussupporting clamps. Too many clamps, causing stresses which are liable to break the apparatus, are just as bad as too few clamps.



Fig. 1. Frame for setting up a free-standing experimental apparatus (measurements in cm.).

#### Glass

The important types of glass used in chemical work are shown in Table 1.

The chemical composition of the more frequently used types of glass is shown in Table 2.

The ordinary starting material for the manufacture of laboratory glassware and connectors consists of glass tubes of circular cross section. The tubing is designated as hand-drawn or machinedrawn; the size reproducibility of the machine-drawn tubing is considerably superior.

Glassware is identified by a special brand number and by the trademark of the firm manufacturing it. A helpful characteristic

Type of glass	Linear coefficient of expansion
Flint glass (Kimble) Pyrex glass Vycor glass Quartz glass	$\begin{array}{c} 93 \cdot 10^{-7} \ (25^{\circ}\text{C}) \\ 33 \cdot 10^{-7} \ (0-300^{\circ}) \\ 8 \cdot 10^{-7} \ (0-300^{\circ}) \\ 5 \cdot 10^{-7} \ (0-300^{\circ}) \end{array}$

Table 1

is the color of the glass, the "hue," which can clearly be seen by transmitted light on a freshly broken end piece. The most common colors vary from yellow to green.

#### Table 2

#### Chemical Composition of Some Types of Glass

		SiO₂	<b>B</b> ₂O₃	Na₂O	K₂O	CaO	BaO	MgO	Al <sub>2</sub> O3	Fe <sub>2</sub> O <sub>3</sub>
Soft	Flint glass (R-6) (Kimble Glass Co., Toledo, Ohio)	67.7	1.5	15.6	0.6	5.6	2.0	4.0	2.8	
	KG-33 (Kimble)	80	13	4	<0.1	<0.1			2	<0.1
Hard	Pyrex (Corning Glass Co., Corning, N. Y.)	80	13	4	<0.1	<0.1			2	<0.1
	Vycor (Corning)	96	3							

In terms of its chemical resistance to attack by aqueous solutions, laboratory glasses are generally classified according to (a) hydrolytic resistance, (b) resistance to acids and (c) resistance to alkali, as shown in Tables 3 and 4.

Many more details about the types of glass can be found in the descriptive literature of the manufacturers.

The various parts of a glass apparatus are assembled into a unit by using ground glass joints, rubber tubing, stoppers, adhesives and especially by sealing glass tubing together with hand torches. The handling of these torches can be easily learned even by one having no previous knowledge of glass blowing. A glass seal

#### Table 3

#### Hydrolytic Resistance

Glass	Conditions	Time	ml. of 0.02 N HCl/10 g. of powdered glass	Weight loss, mg./cm. <sup>2</sup>
Flint glass (Kimble)	steam at 121°C	30 min.	7.8	
Pyrex glass	water at 80°C	48 hr.		0.002
KG – 33 (Kimble)	steam at 121°C	30 min.	0.26	
Vycor	water at 80°C	48 hr.		negligible

at the ends of two glass tubes often can be formed in a shorter time than is required for careful connection of the tubes with rubber tubing. The technique of glass blowing is best learned under the tutorship of an experienced individual; a description of manipulations can thus be omitted here. However, a few hints will be offered:

1. Use glass tubing and other necessary glass from the same manufacturer.

2. Protect glass from dust and store it horizontally; if it is necessary to store it vertically due to lack of space, cover the openings.

3. Before using, clean the glass tubing by pushing or blowing through a moist piece of cotton; clean tubes of larger diameters with a moist rag pulled through on a string; never clean the interior surfaces of glass tubing with an iron or steel wire or another piece of glass tubing. Ignoring this rule is a common cause of cracked tubing during heating.

4. Only freshly cut surfaces, not touched by fingers, should be sealed. When it is impossible to trim an end piece in order to obtain a freshly cut surface, heat the area with a torch and pull off some glass with the aid of a glass rod, or melt the glass, blow this area into a thin-wall bubble and strip it off.

5. When working with hard borosilicate glass (Pyrex), oxygen is added to the air stream through a tee-connector tube.\* The difficulty of working at higher temperatures notwithstanding,

<sup>\*</sup>Blowtorches and hand torches equipped with a valve for oxygen addition are commercially available.

borosilicate glasses are more amenable to glass blowing than the soft glasses because they are much less likely to crack when unevenly heated.

Table 4	able	4
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Glass Conditions		Time	Weight loss, mg./cm. <sup>2</sup>			
Acid resistance						
Pyrex glass Vycor	5% HCl at 100°C 5% HCl at 100°C	24 hr. 24 hr.	0.0045 0.0005			
Base Resistance						
Pyrex glass Vycor	5% NaOH at 100°C 5% NaOH at 100°C	6 hr. 6 hr.	1.4 0.9			

Industrial fusion of pure quartz yields clear quartz glass or vitreous silica. It has the following advantages: low temperature coefficient of expansion, transparency and relatively good, but strongly selective chemical resistance. Tubing, ground joints, etc., of quartz glass can also be made in the laboratory. Oxyhydrogen or hydrogen-air flames with additional oxygen are used. In a pinch, a small industrial oxy-acetylene welding torch will suffice. Despite the high softening temperature of 1500°C, manipulation of quartz is no more difficult than that of ordinary glass. However, the following hints will be useful for those working with quartz glass:

1. Holes often do not close completely in the molten glass; fine capillaries usually remain open. Such spots must be repeatedly remelted or drawn together with a thin quartz rod.

2. Since SiO and SiO<sub>2</sub> vaporize, quartz glass becomes cloudy in the melted area. Remedy: After completing the main sealing operation, remelt the whole area until it is clear, using a large but not too hot oxy-hydrogen flame; if necessary, follow with a rinse of dilute hydrofluoric acid.

3. Rapid blowing is essential because the viscosity tends to increase rapidly on cooling; blowing is best done with a rubber tube.

4. On cooling or on prolonged exposure to heat, there exists the danger of devitrification; that is, conversion of the metastable, glassy form to cristobalite may occur. Once it has started, this process rapidly leads to mechanical failure of the apparatus. The failure starts preferentially at the externally adhering impurity centers and proceeds very rapidly, especially at temperatures in excess of  $1000^{\circ}$ C. Consequently, those parts of quartz glassware which are to be heated and which have already been thoroughly cleaned (with aqueous solutions or organic liquids such as alcohol, acetone, etc.) must not be touched prior to heating because perspiration (NaCl) acts as a devitrifying agent.

The upper temperature limit, when quartz glass is used in the absence of a pressure differential, is  $1250^{\circ}$ C. Unfortunately, evacuated quartz glass flasks start to deform in the  $1150^{\circ}$ C region. The devitrification and warping phenomena make quartz glass vessels unsuitable for experiments in which they must be exposed to temperatures higher than  $1000^{\circ}$ C over long periods of time.

Glasses which cannot be directly sealed together can be interconnected by means of graded seals. Seals having diameters of 7-9 mm. (O.D.) are commercially available. They consist of a series of very short tubes, each with a slightly different coefficient of expansion. In this way, even soft glass can be connected to quartz glass.

Sealing wires into glass is described in detail elsewhere [2]. With quartz glass only molybdenum can be used.

Cleaning of glassware: Glass equipment is usually cleaned with  $CrO_3-H_2SO_4$  cleaning solution by allowing it to stand in the solution for some time, and then rinsing with water. Laug [2] cautions, however, that the glass absorbs  $CrO_3$  upon treatment with this cleaning solution. The  $CrO_3$  cannot be completely removed, even by boiling with water. According to Laug, one gram of glass takes up about 5 mg. of  $CrO_3$ , of which 0.2 mg. remains in the glass even after repeated boiling with water. In certain cases, it is preferable to clean the glassware with concentrated nitric acid. Treatment with alkaline permanganate solution, followed by successive rinsing with water, concentrated hydrochloric acid, and again with water is also very effective.

Glass tubing and apparatus parts which cannot be placed in a drying oven because of their size should not be dried by rinsing with organic solvents (alcohol-ether, acetone); such solvents are often contaminated with low-volatility impurities and these, if left on the glass walls, will cause trouble with sensitive substances, or at high vacuum. Instead, room air should be drawn through the tubes or apparatus by means of an aspirator, with only one opening accessible to the air. This opening should be protected against dust with a cotton wad or a piece of soft filter paper.

Apparatus that is to be taken apart should be provided with ground glass connections. One can use for this purpose standard tapered joints or ball joints. The latter are now manufactured with great precision and are being used more and more. In many cases flanged ground-face connections are advantageous (for details see Mönch [2]). The great advantage of ball joints is their flexibility and easy detachability; they are held together by simple clamps. Their price, on the other hand, is greater than that of the corresponding tapered joints. Ball joints designation includes the diameter of the tube. The following sizes are available on the market:

#### 

In addition, the smallest size, with a ball 12 mm. in diameter, is available with capillaries of 1-3 mm.

The designation of the tapered joints has been changed several times. Table 5 lists the present standards for the different series. All joints are ground with a taper of 1:10 [(larger diameter minus smaller diameter): length of ground portion = 1:10].

The question of which part of the apparatus should carry the male joint, and which the female, is often hard to decide. The best general advice that can be given is to keep the reagents free from contamination. Thus, if the ground joint is to be greased, the female should be on top and the male below; in this case, however, cleaning of the joint is usually more difficult. A groove formed in the ground surface of the male ("two-zone grinding") is very useful in preventing penetration of the grease into the apparatus. Parts which are to be weighed on an analytical balance should carry the male, because it can be cleaned more easily. It is highly recommended that small hooks be attached to both parts of the joint, so that the latter may be held together with springs or rubber bands.

If joints of different materials are to be assembled and heat is to be applied, the female should always be made of the material with the higher expansion coefficient. This applies especially to glass-quartz joints. In an assembly consisting of a glass male and a quartz female, the latter will, as a rule, crack on immersion in boiling water.

Greasing of stopcocks and other ground joints, as well as suitable lubricants and adhesives, will be discussed later. In some cases, it is advantageous to make the connections by cementing and without using any ground joints. This method is especially useful when very large tubes are to be connected, since such cemented seals, if correctly prepared, can be removed without shifting the other parts of the apparatus. The seal is made with a glass sleeve, as shown in Fig. 2. It is best to polish the two butting edges (so that the cut on each is straight) and to interpose a narrow, annealed copper ring, especially if the apparatus is to be evacuated; otherwise, the glass edges may splinter due to the compressive force of atmospheric pressure. To secure sufficient adhesive strength, it is important that the cement be melted by warming the supporting glass. This is especially important with metal cements, since in this case leaks cannot be easily detected. To heat the places to be

#### Table 5

Long	Medium	Short
5/20 $7/25$ $10/30$ $12/30$ $14/35$ $19/38$ $24/40$ $29/42$ $34/45$ $40/50$ $45/50$ $50/50$ $55/50$ $60/50$	5/12 7/15 10/18 12/18 14/20 19/22 24/25 29/26 34/28 40/35	12/10 14/10 19/10 24/12 29/12 34/12 40/12 45/12 50/12 55/12 60/12 71/15
71/60		

#### Designation and Measurements of American Standard Taper Ground Joints (CS 21-39)\*

\* The first number in the designation indicates the larger diameter of the ground section; the second, the length of the ground section.

cemented, one can use a small pilot flame, 10-15 mm. long, created by a glass or metal tip.

If certain precautions are taken, metals can be easily and tightly sealed to glass. This is especially true of Kovar tubing, which can be sealed to Pyrex glass.



Fig. 2. Cementing large glass tubes

With rubber hose connections the edges of the glass tube should be fire-polished. If this is not done, small rubber particles may be scraped off and jammed between the hose and the glass wall, causing a leak. If the hose is lubricated with silicone grease, instead of glycerol or oil, it will not stick. "Frozen" rubber hoses should be cut off. Losing a piece of rubber tubing is preferable to breaking the glass. If one should have occasionally to remove a thermometer (or similar device) stuck in a valuable large rubber stopper, a cork
borer, well lubricated with glycerol, should be introduced between glass tube and stopper and the borer retracted several times, while adding more glycerol.

If rubber stoppers are to be bored, the borer should never be turned in one direction only; instead, the direction should be changed after each half turn, withdrawing the borer several times in order to add more glycerol. Otherwise, the hole gets continually narrower, since the rubber core inside the borer also turns. The hole is then not cut by the sharp edge of the borer but, instead, the rubber is torn out.

# **Ceramic Materials**

The refractory ceramic materials used in the laboratory can be classified, as in Table 6, according to their properties and main ingredients. Unlike glass vessels, their shaping is finished before the high-temperature treatment (firing). Only limited subsequent treatment is possible and this is restricted to mechanical modification (grinding, cutting). Since firing is accompanied by shrinkage, close tolerances can be maintained to a limited extent only. These characteristics restrict ceramic laboratory ware to certain, usually standardized items, e.g., straight tubes, rods, crucibles, dishes, boats, etc.

Group 1. These materials, which consist essentially of  $Al_2O_3$ and  $SiO_2$ , are resistant to extended heating at higher temperatures, but are often not as gas-tight as pure  $SiO_2$ , although some of them come close in this respect. Gas permeability depends very much on the temperature and increases with rising temperature. In addition to the well-known laboratory porcelain ware, some manufacturers have developed special items which have higher chemical or temperature resistances (cf. synopsis in Table 11). The maximum use temperature for these materials increases with the  $Al_2O_3$ content. Again, because of the typical ceramic method of manufacure of these materials (shaping, firing), only some, usually standardized, laboratory items can be made (straight tubes, rods, crucibles, dishes, boats, etc.). Glazes are applied only to porcelain. Ability to withstand temperature changes is much lower than with pure silica.

Chemical resistance at high temperatures is poorest toward alkaline and strongly reducing materials (e.g., active metals). Again, chemical and thermal resistance increases in proportion to the  $Al_2O_3$  content.

For special purposes (e.g., high chemcial resistance), materials of Group 1 can be lined with substances which by themselves are not suitable for ceramic manufacture (for example, MgO, CaO). For example, according to Goehrens [3], one can apply to the vessel a paste made of a mixture of finely ground, weakly ignited and

#### PREPARATIVE METHODS

#### Table 6

Group	Body composition	Designation				
	Dense					
1 a	Aluminum silicate (mullite) (silicate vitreous bond)	Lab. hard procelain $(20-30\% \text{ Al}_2\text{O}_3)$				
1 b	As 1a, with special refrac- tory additions, e.g., corundum, sillimanite, and others	Sillimanite 10a and other special porcelains				
2	Sintered oxides with high m.p., "single phase materials" $Al_2O_3$ , MgO, BeO, ZrO <sub>2</sub> , ThO <sub>2</sub>	Sintered alumina, magnesia, beryllia, zirconia and thoria				
	Porous	8				
3	Same as 1, but less strong vitreous bond; partly pure oxides	Fire clays, mullite, sillimantine, corun- dum (kaolin-bonded)				
4	Carbon	Electrode carbon, retort graphite, graphite (clay- bonded)				

coarse, strongly ignited magnesia in a saturated  $MgCl_2$  solution. This is then transformed by drying and gradual heating into a welladhering protective layer of MgO. In order to deposit a CaO layer (which, among others, can also be applied to ferrous vessels) calcium oxide is made into a paste with calcium nitrate; or, according to W. Jander [3], a paste of CaO and water is painted on to a thickness of 0.3-0.4 cm. Drying and subsequent heating should start at  $40^{\circ}C$  and be increased very slowly up to red heat.

Group 2. For work at very high temperatures, reaction vessels made of ceramic oxide compounds have proved especially suitable; this refers to vessels which have been made by sintering oxides of high purity and of very high melting point. Such materials excel in their resistance to high temperatures and in their remarkable tolerance of a wide range of materials at high temperatures. For almost every material to be melted there can be found an especially suitable ceramic oxide material, as is shown below. Because of the difficulties encountered in ceramic manufacture, the best thermal and chemical resistance characteristics can be achieved only at some sacrifice of flexibility in the choice of ceramic shapes. In the following tables (7-11), which summarize the available practical experience and offer some suggestions for use, the meanings of the symbols are: +++ not attacked; ++ very slightly attacked; + slightly attacked; - strongly attacked; - very strongly attacked; --- completely destroyed.

In using the physical technique of vapor deposition of thin surface layers, some knowledge has been gathered about compatibility between the boat and crucible materials and the reagents heated in these vessels (cf. Auwärter [4]). Table 12 summarizes these data.

Group 3. Besides the materials of Groups 1 and 2, porous ceramics are important. These often are more resistant to

#### Table 7

Behavior of Ceramic Oxide Apparatus with Fused Metals

Metal	°C	Al <sub>2</sub> O <sub>3</sub>	ZrO <sub>2</sub>	MgO	BeO
Li(H2)*	700		++*		
Na(H <sub>2</sub> )	700	+++	+++		+++
K(H <sub>2</sub> )	800	++	+++		+++
Cu(ox)	1200	+++	+++	+	
Be(H <sub>2</sub> )	1500				+++
$Mg(H_2)$	800	++	+++	+++	
$Ca(H_2)$	1000	+++	-	+	
$Al(H_2)$	1000	+++	+++		
$Si(H_2)$	1600	+++	+++		
$Ti(H_2)$	1800	+	+		
$Zr(H_2)$	1700	—	++	+++	++
$\mathbf{Sb}$	800	+++	+++		
Bi	600	+++	+++		
Cr(ox)	1900				
$Cr(H_2)$	1900	+++	++	++	
Mn(ox)	1600				
$Mn(H_2)$	1600	++	++	+	
Fe(ox)	1600				
Fe(H <sub>2</sub> )	1700	+++	+++	+++**	
Ni	1600	+++	+++	+++**	
Co	1600	+++	+++	+++**	
Pb	600	+++	+++	_+++	
Pt	1700	+++	+++	+++ **	
Au	1100	+++	+++	+++ **	

\* Only after previous coating of the crucible with molten LiF. \*\* Vessels made of impure oxides are less resistant.

#### PREPARATIVE METHODS

#### Table 8

Agent	°C	Al <sub>2</sub> O <sub>3</sub>	ZrO2	BeO
$H_2SO_4$ conc.	338	++	+	
HCl conc.	110	+	++	—
HNO, conc.	122	+	++	+
HF conc.	120	++	+++	——
$H_3PO_4$ conc.		++	++	
NaOH 20%	103	++	- <del>  -</del>	++

# Behavior of Ceramic Oxide Apparatus with Liquids

temperature changes. This latter characteristic is sometimes combined with higher maximum use temperatures. Some of these materials are also available as pastes (insulating compounds). Group 4. In this group, use is made of the extremely high melt-

ing point of carbon, which is usually not reached in practice.

#### Table 9

Behavior of Ceramic Oxide Apparatus with Oxides, Hydroxides, and Carbonates

Agent	°C	Al <sub>2</sub> O <sub>3</sub>	$ZrO_2$	MgO	BeO					
$Na_2O_2$	500	+++	++		++					
NaOH	500	÷++•	+++	+++	++					
KOH	500	+++	++	++	++					
Li <sub>2</sub> CO <sub>3</sub>	1000	+++	++++							
Na <sub>2</sub> CO <sub>3</sub>	1000	+++	+++		+++					
K <sub>2</sub> ČO <sub>3</sub>	1000	+++								
Cu <sub>2</sub> O	1300	+++	+++	+++						
$B_2O_3$	1250	+++			++					
810	∫ 1780	+	++		——					
5102	1900									
PbO	900			+++						
$Sb_2O_3$	850	+++	+++							
$Cr_2O_3$	1900	——			—					
MoO <sub>3</sub>	800	+++	++							
Wo <sub>3</sub>	1600	_	++		—					
$Mn_2O_3$	1600			l						
$Mn_3O_4$	1700	1								
FeO	1500	} Al	l of these	materials	are					
Fe <sub>2</sub> O <sub>3</sub>	1600	destroyed.								
$P_2O_5$	600	/ ++								

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Behavior of Ceramic Oxide Apparatus with Molten Salts

ZrO2	‡	+	**	* *+ +	+ + +	***+		++++	++++	+++++	‡	+++++++++++++++++++++++++++++++++++++++	+	+ + +		+ + +	‡	+++	‡	++++	
Al201	* + +	‡	 		‡		<b>†</b>		<b>‡</b>	‡	+	‡ ‡	‡	+++	‡	++++	+++++	‡	‡	<b>‡</b>	
°C	1200	1300	1400	1750	900	1500	1800	1700	1000	800	1750	1100	1650	500	1550	1300	1300	1300	1300	1300	
Salt	$K_2 SO_4$	CuS	Cu <sub>2</sub> SiO <sub>4</sub>	MgSiO,	CaCl <sub>2</sub>	$CaF_2$	$Ca_3(PO_4)_2$	CaSiO,	SrCl <sub>2</sub>	$Sr(NO_3)_2$	SrSO4	BaCl <sub>2</sub>	BaSO <sub>4</sub>	ZnC1 <sub>2</sub>	ZnSiO <sub>3</sub>	$PbB_2O_4$	PbSiO <sub>3</sub>	PbSO4	PbS	FeS	
ZrO <sub>2</sub>	<b>‡</b> +	‡	<b>‡</b>	<b>‡</b>	‡	‡ +	<b>‡</b>	<b>+</b> +	‡	‡	<b>‡</b>	<b>‡</b> +	*++	‡ +	<b>+</b> +	‡	‡ +	+	‡	‡	
Al <sub>2</sub> O <sub>3</sub>	+ + +	‡	<b>†</b>	<b>‡</b>		‡ + +	‡ +	‡ +	+++++	<b>*</b> + +	<b>‡</b>	‡	I	‡	+++++++++++++++++++++++++++++++++++++++	‡	<b>*</b>	++++	I	‡ +	<b>†</b> +
Ω	800	1300	900	200	1200	800	600	400	800	1200	1300	1150	1200	1000	200	500	1000	800	1000	1200	1200
Salt	LiCI	Li <sub>2</sub> SiO <sub>3</sub>	NaCI	NaCN	NaF	Na <sub>2</sub> MoO <sub>4</sub>	NaNO,	NaNO <sub>2</sub>	NaPO <sub>3</sub>	Na4P207	Na <sub>2</sub> SiO <sub>3</sub>	Na <sub>2</sub> SO <sub>4</sub>	Na <sub>2</sub> SiF <sub>6</sub>	Na <sub>2</sub> B <sub>4</sub> O <sub>7</sub>	Na <sub>2</sub> WO <sub>4</sub>	KHSO4	KCI	KCN	KF	KBO <sub>2</sub>	$K_4 P_2 O_7$

<sup>\*</sup> MgO +++. \*\* BeO ----. \*\*\* MgO ----; BeO ++.

A number of apparatus components made of pure carbon, pure graphite or ceramic bonded carbon are commercially available. The additives, however, cause some reduction in refractive properties as compared to pure carbon.

Smaller utensils can easily be prepared in the laboratory from pieces of pure synthetic carbon or graphite. Tubes, plates, valves and other shapes made of pure graphite, as well as of graphite reinforced with synthetics, are commercially available.

## Metals

Although glass and ceramics are the principal materials for chemical apparatus, metals and their alloys are indispensable for many applications. They are superior to glass and ceramics in their high thermal and electrical conductivity, mechanical properties and in their higher ability to withstand temperature changes. In addition, their specific chemical resistance may be important in certain cases. Thus, reactions with fluorine or free alkali metals require use of metal vessels. Metal vessels are also indispensable for high-pressure work.

COPPER

In addition to applications resulting from its electrical conductivity, copper is very useful as a material for vessels employed in work with fluorine (for details cf. Part II, 3 and 4). Aside from this, copper is frequently used for cooling coils and other heat exchangers. Copper tubing of many different sizes is available on the market. It is annealed before use, allowing it to be easily shaped. Since it hardens again on bending, a second annealing may be necessary. Flexible conduits, e.g., connections to steel cylinders, are best made of thin-walled copper tubing, which can be either soldered (hard or soft solder) to the connecting valves or fused to special glasses. Copper develops fissures when heated to red heat in a hydrogen stream. Seamless tombac (a zinc-copper alloy) tubing (also called "spring-tube," "metal-bellows tubing") is more flexible than copper for this purpose (cf. below, Fig. 50 a). Its flexibility is improved by corrugating the tubing walls. Tombac tubing should not be annealed or joined with hard solder. It is joined either with soft solder, or through a special commercially available threaded fitting. The inside grease coating, applied during manufacture, is removed by rinsing with ether and drawing air through while gently heating.

Where low heat conductivity is desired, German silver or similar alloys are employed (cf. Handbook of Chemistry and Physics, Chemical Rubber Publishing Co., Cleveland, Ohio).

Table 11.	Summary	of	Properties	and	Usefulness
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Material	Maximum use temp., C	Max. allowable temp. under pres- sure of 2 kg /cm <sup>2</sup> .°C	M.p., *C	Gas permeability	Ability to with- stand tempera- ture changes	Average thermal expansion coefficient	Density	Thermal conductivity
Quartz (Vitreosil) (fused silica)	1250		ca. 1500	very good	excellent	10-6	2.1	1.08 kcal./ m. hr. C
glazed Hard porcelain unglazed	1100 1300	ca. 1400	1680	very good	satis- factory	10-6	2.46	at 20°C 1.23 kcal./ m. hr. °C
2005 FR 2107 Fire clay	1500	1540	1850 ca. 1730		fairly good fairly good		2.2	
нк 5	ca. 1750	ca. 1700	ca. 1850		fairly good		2.4	good
R Sillimantine	1600 (1300)	1600	1825	porous	very good			good
60	1600			porous	very good			
Marquardt mass unglazed	1700		1825	porous	fairly good	10-5		
K-mass (high alumina) heat resistant porce- lain	1700		ca. 1800	very good	fairly good	10 <sup>-6</sup>	2.46	at 20°C 1.72 kcal./ m. hr. °C
10 a	ca. 1700		above 1800	im- perme- able to gases	quite good	$\sim 0.5 \cdot 10^{-5}$	2.85	depending on the tem- perature: 2.4 kcal./
H	1700		1850	slightly porous	good	$^{\sim 0.6}_{10^{-5}}$	~2.3	2.0  kcal./ m. hr. $C$
Pythagoras mass (a hard porcelain)	1700	ca. 1700	1820	very good	fairly good	10-5	2.9	1.2 kcal./ m. hr. ^C
Sintered zirconia (ZrSiO <sub>4</sub> )	1750	1600	not cons- tant	slightly porous	good		4	low
			-		· · · · · · · · · · · · · · · · · · ·			

Selection and data in this table were made for the purpose of general orientation only. Their properties cannot therefore be guaran-

# of Some Important Ceramic Materials.

Chemical properties	Behavior toward other materials cited in this table	Range of application	
Begins to devitrify above 1150°C. Resists all acids, except conc. phosphoric (above 300°C) and HF, up to the maximum temperatures of use; attacked by bases and basic oxides	Resistant to all raw materials mentioned here	Vessel material for metallic alloy and acid oxide melts. Attacked by Al, Te, Mg, and Mn at high temperatures.	
Good chemical resistance, espe- cially to acidic fluxes, except HF and H <sub>3</sub> PO <sub>4</sub> ; somewhat attacked by strongly alkaline fluxes		Vessel material for metallic, alloy and salt melts up to 1250°C	
Resistant to basic steel melts fluxes; slightly attacked by acid steel melts Satisfactorily resists alkaline melts and acid fluxes		Vessels for metallic melts (steels)	
Resistant to flue dust and furnace gases; reducing up to 1600 °C, oxidizing up to 1300°C Stable in reducing and oxidizing atmospheres	Not attacked by Al <sub>2</sub> O <sub>3</sub> . Caution required with basic oxides, espe- cially MgO	Protective tubes for tem- perature measurements (thermocouples) and for furnaces	
Similar to porcelain, but more basic		Vessels for metal melts	
Considerably higher chemical re- sistance than that of porcelain; more basic than porcelain		Vessels for metallic, alloy, salt and glass melts	
Good resistance to melts, espec- cially acidic ones Resistant to attack by ash components and flue dust	Compatible with all other materials con- taining alumina and silica	Vessels for metallic and alloy melts (Tamman crucible), protective tubes for pyrom- eters. External protective tubes for pyrometers; sheathing tubes for electric-oven heating	
More resistant than porcelain to all forms of chemical attack. Re- sistant up to 1600°C to most acid fluxes		Vessels for metallic, alloy, salt and glass melts	
Resistant to acids; attacked by basic substances at high temperature	Least compatible with BeO, MgO	Vessels for metallic and alloy melts	

teed and this table should not be interpreted as recommendation of specific products.

# Table 11 (continued)

Material	Maximum use temp., °C	Max. allowable temp. under pres- sure of 2kg /cm <sup>2</sup> , C	М.р., °С	G <b>as</b> permeability	Ability to with- stand tempera- ture changes	Average thermal expansion coefficient	Density	Thermal conductivity
Corundum bonded with kaolin, sin- tered (Al <sub>2</sub> O <sub>3</sub> + 5%SiO <sub>2</sub> )	1800	1700	under 2000	porous	good		3.5	good
Sintered alumina (Al <sub>2</sub> O <sub>3</sub> )	1850, poss. more	1750	2050	good	good	10 <sup>-7</sup> to 10 <sup>-6</sup>	3.4- 3.9	good
Sintered beryllia (BeO)	under 2200	2150	2550	good	very good		2.9	very good
Sintered magnesia (fused MgO)	2200	2000	2700	porous	moderately good	10 <sup>-5</sup>	2.8	good
Sintered zirconia (ZrO <sub>2</sub> )	2500	1900	2700	slightly porous	low	10 <sup>-5</sup>	5.4	low
Sintered thoria (ThO <sub>2</sub> )	2700	1950	3000	good	poor	high thermal expansion coefficient	9.2	low
Carbon	above 3000	none	practi- cally infusi- ble	porous	very good	10-6	ca. 1.5	3.5-8 kcal./ m. hr. °C
Electro- graphite	above 3000	none	practi- cally infusi- ble		very good	10-6	1.5— 1.7	at 20°C 100 kcal./ m. hr. °C

Chemical properties	Behavior toward other materials cited in this table	Range of application
Resistant to alkali, alkali metals and other metals as well as glass and slag fluxes. Not attacked by chlorine, carbon, carbon monox- ide, hydrogen, hydrocarbons, etc., even at highest use tem- perature. Scarcely attacked by even the strong mineral acids, e.g., hydrofluoric or sulfuric	Incompatible with $ZrO_2$ , $ZrSiO_4$ and $BeO$ ; compatible with other oxides. Caution recommended with ThO <sub>2</sub> at temperatures over 1500°C	Vessels for high-melting metallic and alloy melts
Resistant to alkaline materials and to reduction by molten metals, carbon, carbon monoxide and hydrogen. Attacked above 1800°C. Incompatible with SiO <sub>2</sub>	Compatible with ZrO <sub>2</sub> up to about 1850°C; in- incompatible with other oxides	Vessels for high-melting metallic and alloy melts
Resistant to basic materials even at the highest temperatures. Not resistant to (strongly re- ducing) carbon at high tempera- tures	Least compatible with ZrSiO <sub>4</sub>	Vessels for high-melting metallic and alloy melts
At the highest temperatures, re- sistant to a very wide range of acid and basic materials. Car- bide formation with carbon at high temperature	Very poor compati- bility with Al <sub>2</sub> O <sub>3</sub>	Vessels for high-melting metallic and alloy melts
Resistant at extremely high tem- peratures, especially to alkaline materials. Stable to reduction by high-melting metals. Car- bide formation with carbon at high temperatures	Relatively good com- patibility with all oxides; best with ZrO <sub>2</sub>	Vessels for high-melting metallic and alloy melts
Resistant to acid and basic fluxes if these do not oxidize. Some surface contamination when metals are fused in vessels made of this material	Formation of SiC above 1400°C on con- tact with silica- containing materials	Vessels for silicate melts, sinter processes, pro- duction of refractory metals and reduction of metal oxides
Highly resistant; attacked only by oxidizing agents, e.g., air above $550^{\circ}$ C, steam and CO <sub>2</sub> above $900^{\circ}$ C; stable to metals, if these do not form carbides. More suitable for melting experiments than carbon crucibles because less reactive	If the other material contains SiO <sub>2</sub> , silicon carbide is formed above about 1300°C. Stable Al <sub>2</sub> O <sub>3</sub> , BeO, and MgO up to about 1800-1900°C	Same as carbon

## Table 11 (continued)

Material	Maximum use temp., °C	Max. allowable temp. under pres- sure of 2kg/cm <sup>2</sup> , C	M.p., °C	Gas permeability	Ability to with- stand tempera- ture changes	Average thermal expansion coefficient	Density	Thermal conductivity
Silicon carbide (SiC)	1500		1500*	slightly porous	very good	0.5 · 10 <sup>-5</sup>	2.2	8.5–4.8 kcal./ m. hr. °C
Graphite- bonded clay (crucible)	са. 1700	1700- 1800		almost imperm- eable to gases when glazed	cxcel- lent		1.6	very good

#### SILVER

Silver, like copper, is used for its high electrical and thermal conductivity and further, for its resistance to fused alkali. Pure silver crystallizes on extended heating at red heat and becomes brittle. An alloy with 0.1% nickel is free of this drawback.

#### GOLD

Pure gold is too soft for laboratory ware. Gold-platinum alloys are sometimes used for their alkali resistance.

#### PLATINUM GROUP METALS

The metals used include platinum, rhodium, iridium and palladium. Precautionary measures to be taken in handling platinum are well known through the literature circulated by firms producing noble metals (cf. also Part II, 29, Platinum Metals). Rhodium is ordinarily used only in alloys. However, it can also be used for extremely high-melting crucibles, provided appropriate steps are taken to compensate for its tendency to oxidize in air. Platinumrhodium alloys can withstand very high temperatures because their vapor pressure is very low. They can thus be used for heat conductors and thermocouples. Although iridium has an appreciably higher melting point than platinum, its vapor pressure is more than ten times greater. In spite of this, it is suitable in special cases for vessels in which strongly basic oxides, like BaO, are to be heated in an oxygen-containing atmosphere. For example, it was used in the form of a channel heated by direct passage of

Chemical properties	Behavior toward other materials cited in this table	Range of application
Resistant to attack by ash components and flue dust	Good compatibility with alumina and silica materials	Suitable as external protection tube for pyrom- eters due to its good heat conductivity
Attacked by alkaline fluxes. With melts low in carbon and containing Fe or Ni, possibility of carbonization; this can be prevented by an interior lining		Vessels for all metallic melts, except electron metal

current [G. Wagner and H. Binder, Z. anorg. allg. Chem. 297, 328 (1959)]. Platinum-iridium alloys are very hard and can be employed as electrodes in the preparation of ultrapure chlorine by electrolysis of acidic saline solution, provided they contain a sufficiently high percentage of iridium. Palladium is cheaper

## Table 12

Behavior of Some Materials that can be Vaporized in a High Vacuum with Common Crucible Materials at Temperatures Above 1000°C (++ very suitable; + suitable; - unsuitable)

		Crucible material							
He (Vaporiza	eated material tion temperature, °C)	M	Mo	Та	Al <sub>2</sub> O <sub>3</sub>	$ZrO_2$	$SiO_2$	TiC	$\operatorname{ZrB}_2$
Al	(1300)	+	_	_	++	++	-	+	++
Ti	(1800)	+		—				++	
TiO	(1600)	++	—	—				++	
Cr	(1550)	++	+						
Fe	(1550)	+							
Si	(1550)	+							
SiO	(1250)	++	++	++	++	++			
Ge	(1100)	+							
$CeO_2$	(1850)	+	+						
$MgF_2$	(1050)	++	++	++	++	++	++		
ZnS	(1000)	++	++	++	++	++	++		

than platinum and is used as an alloying agent. The high permeability of red-hot Pd to hydrogen is used for the preparation of very pure hydrogen.

# TUNGSTEN, MOLYBDENUM, TANTALUM, NIOBIUM

This group of metals possesses the highest melting points, lowest vapor pressures, high strengths and low coefficients of thermal expansion. They find many uses in the laboratory (tungsten and molybdenum furnaces, seals with quartz and other glasses, etc.). These metals are commercially available in large pieces, sheets, tubes, wires, etc. At higher temperatures a protective atmosphere or a vacuum is absolutely necessary. In the case of Mo and W, the protective blanket may consist either of inert gases or of hydrogen or a hydrogen-nitrogen mixture (synthesis gas). However, only the inert gases should be used for niobium and tantalum. Tantalum has a very high chemical resistance, with special resistance to hydrogen chloride. Molybdenum is stable to free alkali and alkaline earth metals even at high temperatures.

#### IRON AND NICKEL

Their use in the laboratory is well known. Very pure iron (e.g., carbonyl iron) and pure nickel, and sometimes also high-grade alloy steels, serve as crucible and boat materials. In particular, they are resistant to liquid and gaseous alkali and alkaline earth metals at high temperatures.

## JOINING BY WELDING AND SOLDERING

Welded joints are the best for most uses. Platinum group metals are either welded directly in an oxy-hydrogen flame by melting together (thermocouple wires; see below under Thermocouples), or they are heated to bright-red heat and joined by a sharp hammer blow. Welding of other metals should be entrusted to an experienced specialized machine shop (see Angerer [1]).

Hard soldering is applicable in most cases. Spelter solder, silver solder (m.p. about 700°C) and pure silver (m.p. 960°C) are used. The cleaned junctions are sprinkled with a generous amount of borax, and when they are sufficiently hot the solder (as powder or as wire) is added. With a large amount of borax and pure silver, even Mo and W can be hard soldered.

Soft soldering with a lead-tin solder is common. This is usually done with a soldering fluid (a solution of ZnCl<sub>2</sub> in HCl) or soldering paste to deoxidize the junctions. However, a thorough mechanical cleaning of these junctions is also essential. Special solders for aluminum are also commercially available.

## GLASS TO METAL SEALS

The compatibility of Pyrex and Kovar has already been mentioned. Platinum and Pyrex are also compatible as are tungsten and uranium with Nonex glasses. Other metals may be used with special glasses available from the Corning Glass Co., Corning, N. Y. Only those metals whose coefficients of thermal expansion between room temperature and the transition point of glass differ by not more than 10% can be used for making glass to metal seals.

For more information, see [5].

### Plastics

Among the many kinds of plastics, some have secured a permanent place in the laboratory, mainly because their resistance to acids (especially HF) and to alkalis. Plastic tubing is transparent and very durable; its advantage over rubber tubing is that it is rather stiff and does not pinch. Plastic tubing should be slightly preheated before slipping onto a glass tube (by dipping into hot water, for example) and greased with a drop of oil. Thermoplastic materials (Plexiglas) are readily workable by bending (when heated), sawing, turning and cementing. They can be welded with a simple hot-air device constructed for this purpose. In this device, the temperature of the hot air is readily controlled and the welding rod is of the same material as the other parts.

Polyethylene is resistant to strong acids and bases. It is attacked by halogens. It can be used up to 70°C. On cooling with liquid nitrogen it invariably develops fissures. It is joined by heat sealing.

Polyvinyl chloride is resistant to strong acids and bases.

Teflon and Kel-F are especially resistant to boiling concentrated mineral acids, including aqua regia, and to free halogens and most organic solvents. (Teflon = polytetrafluoroethylene; Kel-F = polytrifluorochloroethylene.) While both of these materials are somewhat difficult to form and machine in the laboratory, they can be used up to almost  $300^{\circ}$ C.

# **Pure Solvents**

#### VERY PURE WATER

See also Part II, 1. Many excellent devices for producing single and double distilled water are available. Ion exchange purification is suitable for many purposes. For small-scale conversion a makeshift apparatus of this nature can easily be constructed: e.g., a glass tube 50 cm. long, 3-4 cm. diameter, filled with granulated ion exchange resin. Large-scale apparatus with electronic purity control is commercially available.

#### ALCOHOL

The customary method of dehydration with quicklime yields appr. 99.5% alcohol, which is satisfactory for most purposes. For a purer product, this can be further dehydrated by refluxing with calcium chips, followed by distillation. It is better to perform the regular distillation over lime not in a round-bottom flask but in a copper still, which can be heated in an oil bath to 150°C. Only in this way can the major part of the alcohol used be recovered, since without vigorous heating a considerable amount is retained by the lime. Such a still with a removable head pays for itself after a short time, since glass flasks frequently break when the solidly adhering lime cake is being removed. Even better, alcohol can be heated with lime in an autoclave for 1-2 hours at 100°C, and distilled off by opening the valve. Quick water removal can also be achieved by boiling with calcium carbide (175g. CaC2 per 1000 g. alcohol; reflux 30 min.; add 1 g. CuSO<sub>4</sub>; reflux another 15 min., then distill off. Caution! Use only a water bath for heating; the copper acetylide formed is explosive. The product is almost 99.9% alcohol. Very pure alcohol can also be obtained by dehydration with magnesium. A small amount of lime-dehydrated absolute alcohol is added to a small excess of magnesium turnings in a flask equipped with a reflux condenser. The quantity of turnings (for binding the water) is calculated on the basis of the total amount of alcohol, according to the method of Grignard. Following the addition of a few grains of iodine, the flask is heated to boiling. After the start of the reaction, commercial 96% alcohol is added slowly through the condenser; the alcohol in the flask should, however, never become too diluted with water or the reaction will stop. Finally, the alcohol is distilled off. About 75-100 g. of Mg is required per liter of alcohol. Methanol can, of course, be dehydrated the same way. The troublesome heavy bumping encountered during distillation of alcohol from lime can be avoided if the alcohol is always kept at the boiling point. In other words, after the dehydration is completed, the condenser head should be switched from the reflux to the distilling position while the solution remains boiling. Another efficient dehydration procedure for predried alcohol is refluxing for two hours with an addition of sodium (7 g. /liter) and diethyl phthalate or diethyl succinate (25 g./liter), followed by distillation of the alcohol from the high-boiling ester [Smith, J. Chem. Soc. London 1927, 1288; see also J. Amer. Chem. Soc. 53, 1106 (1931)]. The last method yields a product with a water content of less than 0.01%. For determination of water content, see E.

Eberius, "Water Determination with Karl Fischer Reagent," 2nd edition, Weinheim, 1958.

#### ETHER

Preparation of absolute ether by shaking with  $CaCl_2$ , allowing it to stand over sodium wire and subsequent distilling is well known. Peroxides, which can be easily detected in ether with a titanium(IV)sulfuric acid mixture, are removed by shaking with a solution of 600 g.  $FeSO_4$ , 60 ml. $H_2SO_4$  and 1100 ml.water in a separatory funnel. Separation of the layers is followed by distillation. Equal volumes of ether and solution are used. It is then stored in well-filled bottles over sodium wire.

For the purification of other organic solvents, consult the wellknown textbooks on methods of organic chemistry.\*

#### Mercury

Mechanical impurities are removed by filtration through leather, a glass suction funnel, a porcelain filtering crucible or by the makeshift device of filtration through a paper filter, the apex of which has been pierced several times with a pin.

Dissolved base metals are removed by shaking with oxidizing agents or acids, or by aeration; these processes are preferably combined, as illustrated in Fig. 3. Shaking with 5% mercury nitrate solution containing 15-20% HNO<sub>3</sub>, then with very dilute HNO<sub>3</sub>, and finally with water is also recommended. Recently, treatment with cold, saturated KMnO<sub>4</sub> solution was indicated to be very effective. The mercury should be shaken repeatedly with fresh solution until the color of the KMnO<sub>4</sub> no longer changes over a period of half a minute. It is then washed with water, allowed to settle and acidified with a small amount of HNO<sub>3</sub>; with this treatment the Hg coalesces. It is then washed, dried by heating in vacuum and finally distilled.

Suitable devices for distillation, for example, that shown in Fig. 4, are commercially available. Figures 5 and 6 illustrate types of apparatus readily made from Pyrex glass [cf. D. Goux, Chim. Ind. 70, 216 (1953)]. The apparatus is attached to a suitable stand and a small red control lamp is connected in parallel to several coils of the electric heating element. Once the apparatus is evacuated at F, it will keep on continuously evacuating itself,

<sup>\*</sup>For example, C. Weygand, Organisch-chemische Experimentierkunst [Technique of Experimental Organic Chemistry], Leipzig, 1948.



Fig. 4. Automatic distillation of mercury: K: cooling sleeves of slotted and bent aluminum foil. Fig. 5. Automatic distillation of mercury. (Length of distillation vessel 180 mm., diameter 35 mm.; the vessel is insulated with a thick layer of asbestos.)

should traces of gas enter along with the impure Hg, since tube B, assuming its diameter is not larger than approximately 2 mm., acts as the down pipe of a Sprengel pump.

Pure Hg should leave behind no "tail" on decanting. Mercury tongs or a mercury pipette (Fig. 7) may be used to pick up spilled Hg. The pipette is operated with an attached vacuum pump (water aspirator). With this device, spilled Hg can also be retrieved from cracks.

# **Sealing Materials and Lubricants**

The choice of sealing materials and lubricants deserves particular consideration, especially since a great number of





Fig. 6. Automatic distillation of mercury.

Fig. 7. Mercury pipette.

suitable substances are available today to meet even specialized requirements.

LUBRICANTS

These are principally used for ground joints and stopcocks. Numerous commercial products are available; of these, the following are most frequently used:

Ramsay grease has many uses and is commercially available in two forms: "viscous," chiefly for standard stopcocks and ground joints, and "soft," for large stopcocks and ground joints as well as desiccators, and for use at lower temperatures. This lubricant satisfies most of the requirements of preparative laboratory technique and even suffices for high-vacuum work. It can be prepared by mixing paraffin, vaseline and crude rubber (1:3:7 up to 1:8:16).

Apiezon greases are rather expensive, but indispensable for the most stringent conditions of high-vacuum work. Their vapor pressure is immeasurably low at room temperature. They are also rather resistant to halogens but, because of their greater fluidity, have the disadvantage of being more easily squeezed out of the lubricated surfaces when used for large stopcocks and ground joints. This can be prevented if a band of Ramsay grease is placed at the upper part of the stopcock or ground joint. The Ramsay grease has less tendency to flow because of its rubber content. In this case, care must be taken that the Ramsay grease does not get into the apparatus to be evacuated. Apiezon grease is commercially available in two consistencies, P and R; P is the most widely used.

Silicone grease, which is chemically very resistant, is also recommended to prevent rubber from sticking to glass. Its vapor pressure at room temperature is immeasurably low. It is also serviceable at rather high temperatures. The author has observed that, with this grease, stopcocks that have not been used for a rather long time have a tendency to stick. When warmed, however, they can almost always be readily loosened again.

Greaseless lubricants. If ground joints or stopcocks come into contact with organic solvents, the use of the previously mentioned lubricants is inadvisable. In these cases, a mixture of melted sugar and glycerol can be useful. Kapsenberg recommends triturating 25-35 g. of dextrin in a porcelain dish with 35 ml. of glycerol, added gradually, and then heating the mixture over a free flame, with stirring, until a grease of honeylike consistency is formed. This is heated twice until it foams and is then filtered through cotton wool. It should be stored in a glass-stoppered bottle. It is hygroscopic and somewhat more viscous than vaseline. A paste made from very fine bentonite with glycerol is frequently useful.

Stopcock greases stable to chlorine can be obtained by chlorination of paraffin-stearin mixtures at 150°C and additional treatment with NOC1. The chlorinated mixtures are degassed by heating in vacuum. At higher temperatures perchloronaphthalene may also be used as a lubricant. Apiezon greases are also fairly stable to chlorine, even without preliminary treatment.

#### REMOVABLE CEMENTS

Suitable cements should have low vapor pressures and should not be too brittle.

Picein, vapor pressure approximately  $10^{-4}$  mm. ( $20^{\circ}$ C), is useful. It may be used up to approximately  $60^{\circ}$ C, and is readily soluble in benzene and toluene. Other waxes with low vapor pressures and variable hardnesses and usable up to  $80^{\circ}$ C are available from the J. G. Biddle Co., Philadelphia.

In place of the opaque, black picein, clear and transparent polyvinyl acetate may also be employed. Those polyvinyl acetates which soften at a low temperature are used in a manner similar to picein. It should be noted, however, that polyvinyl acetate chars rather easily when in contact with a free flame. Polyvinyl acetate is insoluble in water and aliphatic hydrocarbons but is soluble in esters, ketones, chlorinated hydrocarbons and benzene. Polyethylene, in the form of a film placed between the previously heated surfaces of a ground joint, is especially suited as a sealing material for joints used at higher temperatures.

#### CEMENTS FOR HIGHER TEMPERATURES

Silver chloride, melting point 455°C, adheres excellently to glass, quartz and metals, but only if (according to Stasiw and also Von Wartenberg) a few small granules of Ag<sub>a</sub>O have been dissolved in the previously fused AgCl; the Ag<sub>2</sub>O is dissolved essentially without decomposition. Mönch recommends lowering the melting point by addition of TlCl. A mixture of 27.2 g. of TlCl and 18.2 g. of AgCl melts at 210°C. A mixture of 3 g. of TlCl, 4 g. of AgCl and 6 g. of AgI melts still lower (131°C) [R. O. Herzog and H. M. Spurlich, Z. physik. Chem. (Bodenstein Anniversary Volume), 241 (1931)]. Alloys. Wood's metal, Rose's metal (see section on Alloys). Alloys of 40 parts Bi, 15 parts Hg, 25 parts Pb, 10 parts Sn and 10 parts Cd adhere especially well to glass. Pure indium metal (m.p. 155°C) and various indium alloys (for example, 50% In +50% Sn, m.p. 117°C) are suitable for joining metal to glass, quartz or ceramics. The surfaces of the parts must be very clean. Precautions should be taken with regard to the temperature ranges suitable for the various allow cements and for the materials to be cemented.

#### PERMANENT CEMENTS

Glycerol-litharge cement. Glycerol is dehydrated as completely as possible by heating at a high temperature; litharge is likewise heated at 200 to 400°C. After cooling, 20 g. of litharge is stirred with 5 ml. of the anhydrous glycerol. The surfaces to be cemented are rubbed beforehand with glycerol. Setting time, approximately 1/2 hour. The cement withstands temperatures up to approximately  $300^{\circ}$ C. The cemented spots can be loosened with a strong sodium hydroxide solution.

Waterglass cements. Mixtures of feldspar and waterglass or of talc and waterglass are serviceable up to quite high temperatures. The two components are stirred together to form a thick paste and the cemented parts are then first allowed to dry in the air and later, slowly in the drying oven. The cement withstands quite high temperatures.

Zinc oxide cements. Zinc oxide, stirred with zinc chloride solution, hardens in a few minutes to a stonelike mass. Dental cement (obtainable from dental supply houses) also belongs to the class of zinc oxide cements; it consists of a solid and a liquid component and after trituration hardens in a few minutes. The fact that the volume remains constant on hardening is especially advantageous. Bakelite cements. Bakelite lacquer is used as a cement and the cement is then heated in the drying oven. In this way, solidification to a very hard mass takes place. Mixtures of Bakelite with talc, prepared chalk or kaolin may also be used as cements.

Epoxy resins, when mixed with a hardening agent, cross link on gentle heating; in this way, very strong and even vacuum-tight joints between the following materials are obtained: metal, glass, porcelain, thermosetting (not thermoplastic) synthetics, vulcanized rubber. Epoxy cements can be cooled to very low temperatures without cracking and have very low vapor pressures  $(10^{-6} \text{ to } 10^{-7} \text{ mm.}$ at room temperature.

In conclusion, various commercially available household cements may be mentioned. It is not possible to enumerate all of these; however, they often prove to be very useful in the laboratory.

# **High Temperatures**

#### COMBUSTION HEATING

Except in unusual circumstances, only gas burners need be considered for the laboratory; these are commercially available in a great variety of well-known types. With these burners, small crucibles may be heated to approximately 700-800°C, and using a Winkler clay forge, even to approximately 100°C higher. For still higher temperatures, the well-known blast burners are used; the compressed air necessary for their operation is produced either by a water or an electrically operated compressor. A very hot flame is produced by admixing O<sub>2</sub> to the blast in a mixing tee. Highly recommended blast burners with finely adjustable auxiliary connections for oxygen are also commercially available. It is not necessary here to go further into the subject of the numerous types of gas furnaces, of which the Rössler furnace is the best known. Furnaces based on the Schnabel principle of flameless combustion on thermostable packing material are very effective. Figure 8 schematically illustrates the construction of such a furnace (J. D'Ans, E. Ryschkewitsch, T. Diekmann and E. Houdremont [6]). With petroleum-oxygen mixtures, very high temperatures (up to  $2600^{\circ}$ C) can also be attained (H. von Wartenberg [6]).

Acetylene hardly needs to be considered for use in laboratory burners. On the other hand, special furnaces with oxy-acetylene burners reach very high temperatures (up to 3200°C).

## ELECTRICAL HEATING

Electrical heating apparatus is becoming more and more popular in the laboratory, even for purposes for which only Bunsen burners were previously used. Hot plates, flask heaters, etc., are offered by all distributors of laboratory supplies, but simple household hot plates are also frequently used. Unlike Bunsen burners, electrical heaters provide steady heat, not interrupted by occasional air draft, and may be controlled by small commercially available and inexpensive regulators. The latter are based on periodic current interruption, the timing of which is controlled with a dial knob. Naturally, one can also make use of transformers or rheostats; however, the latter cause power losses. Various types of immersion heaters, which for chemical work are available sheathed in quartz glass, must be especially mentioned.



Fig. 8. Furnace for flameless combustion.

For rapid, loss-free surface evaporation of liquids, quartz heaters, also called surface irradiators, are used. Various electrically heated water baths and air baths are also on the market. Air baths in which the heating elements do not reach red heat are also assigned to the infrared heater class. They reduce the possibility of igniting highly flammable fumes. Heating units in the form of cushions, hoods or tape made from glass fabric with embedded heating wire are available. They are known as "mushroom" heating hoods or ''electrothermal'' mantles. These are especially useful for heating flasks or tubing filled with flammable liquids.

Many well-designed electrical crucible furnaces are priced so low and are so well known that detailed discussion of these is superfluous. Electrical furnaces for heating tubes are often made right in the laboratory because they have to be frequently adapted for special purposes, for which adequate equipment is not commercially always available. Loss of current and material can be avoided by using the correct dimensions. Good thermal insulation is particularly important, not only to save current, but for the workers' comfort, especially during the hot summer months. The following furnaces are classified according to type of heating element:

1. Wire-wound furnaces.

2. Silicon carbide rod or tubular furnaces.

3. Carbon (graphite) tubular furnaces.

4. Special furnaces: iridium and tungsten wire furnaces, highfrequency heating furnaces, cathode ray ovens, arc furnaces, etc.

# WIRE-WOUND FURNACES

The heating element consists of an alloy conductor in either wire or tape form (Nichrome, Kanthal, Megapyr, etc.). The conductor can also be platinum wire or tape or molybdenum wire.

Platinum-wound furnaces are commercially available. They are especially useful in specialized small units where constant high temperature is required. In such cases the furnaces are internally wound. The making of such furnaces is further described below. Furnaces with Nichrome, Kanthal, Megapyr, etc., elements are constructed as follows: After the size of the furnace for the intended application is determined, the tube on which the heating element is wound is selected. For temperatures below 500°C aluminum tubes are satisfactory. Steel tubes can be used up to 600-700°C. For higher temperatures only ceramic tubes are acceptable. Metal tubes help ensure even distribution of temperature throughout the furnace. Unglazed porcelain, Pythagoras mass, K-mass, Sillimanite, Sillimantine and sintered alumina can be used as ceramic tubes. The use of alumina is reserved for especially high temperatures. Construction of a furnace is simple when threaded tubes, on which the conductor is wound, are used. According to R. Fricke and F. R. Meyer [6], very neat furnaces, with the additional advantage of transparency, can be made from pieces of glass tubing (Pyrex, Vycor). These fine furnaces are restricted to the lower range of high temperatures. The tube is continuously wound with the conductor, which is held in place by the tightness of the winding.

To determine the length and cross section of the conductor, the surface of the tube is first measured. Then the wattage needed for reaching the desired temperature (assuming moderately good insulation) is estimated according to the following empirical rules.

Up to 300°C, 20 watts per dm.<sup>2\*</sup> are required; for each 100° increment up to 700°C, 20 additional watts per dm.<sup>2</sup> over the basic figure; therefore, 100 watts/dm.<sup>2</sup> for 700°C. Between 700° and 1100°C, 30 watts are necessary for each additional 100°. Between 1100° and 1300°C: 40 watts. Between 1300° and 1500°C: 50 watts. Above 1600°C: 60 watts; accordingly for 200°C, 700 watts/dm.<sup>2</sup> are required. The required amperage is then calculated from the available voltage. As a safety factor, the voltage figure should be reduced by 10%. Resistance in ohms and approximate wire length are then calculated. For smaller furnaces the distance between spirals is held to about 1 mm. Thus, about 2 mm. of tube length is required per turn. The wire length is calculated from the tube circumference. As most wire material has a resistance of 1 ohm/mm<sup>2</sup> of cross section, an approximate figure for the cross section of the wire can now be determined. This figure is sufficiently accurate for laboratory purposes.

Sample calculations for a small laboratory tube furnace: the furnace is to reach  $900^{\circ}$ C; the tube diameter is to be 2 cm., the tube length 30 cm.; the available power supply is 220 volts. As mentioned before, the latter should be reduced by a factor of 10%. Therefore, the calculations are based on 200 volts, in order to assure the attainment of the required temperature, as well as permit some temperature regulation. The surface to be heated is about 2 dm<sup>2</sup>, and using the aforementioned empirical rule, 320 watts is required for 900°C. At 200 volts, 1.6 amp. is necessary. For 1.6 amp. at 200 volts the conductor resistance must be 200/1.6 = 125 ohms. At a 2-mm, coil pitch and a 30-cm, tube length we arrive at 150 turns, or, at a circumference of 6 cm. a wire length of 9 m. The maximum load of the usual heating wire (Nichrome), 0.6 amp. for each 0.1 mm, of wire diameter, is normally assumed. Thus, a conductor of 1-mm, diameter will carry 6 amp. The resistance of such heating wire is generally indicated on the spool (Nichrome about 1.3 ohms/mm<sup>2</sup>; see also the Handbook of Chemistry and Physics, Chemical Rubber Pub.) For wire of 0.3-mm. diameter, a wire length of 9 m. is required at a coil pitch of 2 mm.

If platinum wire is chosen as the conductor, the high temperature coefficient of resistance, which at 1000°C is 3 to 4 times that at room temperature, must be taken into account. Therefore, platinum furnaces should always be heated slowly, using a rheostat in series with the winding. Otherwise the furnace is in danger of burning out at the hot spots, since the heat transfer from the wire to the furnace wall is not uniform. When the length of the conductor has been determined, the winding of the wire can start. For use below 1000°C it is advantageous to first wrap the tube with a layer of moist asbestos paper. When that has dried, the wire is wound on top of it. In high-temperature furnaces the wire is wound directly around the tube. The wire cross section is increased at the ends by twisting the conductor around itself and the wire is fixed at each end by a loop or, even better, by a sleeve slipped over the tube. Both ends of the winding are secured in the same manner. In furnaces used up to 1000°C, a talc-waterglass paste is applied in a layer approximately 1 mm, thick over the surface of the conductor. For higher temperatures, an aqueous paste, made of

equal parts of carbonate-free MgO and silica-free Al<sub>2</sub>O<sub>3</sub>, can be used (insulating compounds containing free silicic acid destroy the conductor rather quickly at higher temperatures). After air drying, the tube is dried in an oven and finally heated by passing current through the winding. Now the tube can be placed in a pipe or a sheet metal housing. The free space between the tube and the housing is filled with magnesia or diatomaceous earth. Ready made, easy to handle magnesia (plus additive) insulating material is commercially available. Adequate furnace insulation can also be easily made from asbestos pulp, cemented together with waterglass. The free spaces are filled with mineral wool, loose asbestos or MgO, if necessary in layers (MgO inside, mineral wool outside). For low temperatures, wrapping with several layers of asbestos cord is sufficient. The protruding wire ends leading to terminals are insulated with ceramic insulating beads (available from electric supply houses). A ribbon conductor can be advantageously used instead of wire. For even distribution of temperature in the furnace, we recommend closer winding at the ends than in the middle, as the ends always tend to be cooler than the middle of the furnace. However, it is difficult to find the right coil pitch without some careful experimentation. Therefore, it is sometimes desirable to add supplementary windings near the ends and control them separately.

Furnaces with uniform temperature distribution over the whole length of the tube, including the furnace ends, can be made from a single block of aluminum or bronze (wall thickness 15-20 mm. with a longitudinal hole drilled for a thermocouple). Heating wire, strung with insulating beads, is wound around the block.

The heating coil branches off at about 5-10 cm. from each end of the furnace. Each branch consists of a twisted wire, connected to the main coil. The middle portion of the coil, in which the current must be lower than in the end sections, is thus isolated. The branch wires can be connected via a suitable rheostat, thus regulating the current in the newly formed parallel circuit. Very good insulation, projecting over the block ends, is mandatory.

Stands equipped with mechanisms for raising and tilting the furnace are excellent for mounting purposes. This type of mounting also permits the furnace to operate in an inclined position (see Fig. 9).

#### FURNACES WITH INTERNAL HEATING COILS

These are not as difficult to make as it would seem at a first glance. With Pt wire, the wire length must be calculated based on the resistance at maximum temperature. A round wooden core, with a diameter 1-2 mm. smaller than that of the heating coil, is turned out on a lathe. This core is then cut lengthwise into three wedges as shown in Fig. 10. The parts are reassembled and wrapped

with a layer of tissue paper, and a thin string is tightly wound along the whole length of the assembly. The assembly is then wrapped with a few additional layers of tissue paper, the paper is lightly soaked with oil, and the assembly is finally wound with Pt wire. The wound assembly is coated with a waterdispersed ceramic powder and allowed to dry. It is recoated after drying and inserted into a suitable porcelain tube while still moist. Any free space is filled with insulating compound. After air drying, it should be dried thoroughly in an

oven. The string is then carefully pulled out and the wooden core removed by extracting the middle wedge. Again the coreless assembly is thoroughly dried, and then slowly and cautiously heated

until the tissue paper has been incinerated. After cooling, the inside of the furnace is coated with insulating compound, air dried, and then carefully baked until complete dryness. A heating coil can also be embedded in thermal insulation in the same manner as described above. Such furnaces can be used up to 1500°C

without difficulty, whereas externally wound Pt furnaces cannot withstand temperatures above 1250°C for any length of time. Rhodium alloys should be used for higher temperatures.

## MOLYBDENUM WIRE FURNACES

These can be used up to  $1500^{\circ}$ C. However, the heating coil must be protected against burnout by a constant, slow flow of protective gas (H<sub>2</sub>, water gas, i.e., CO +H<sub>2</sub>, or N<sub>2</sub> +H<sub>2</sub>). These furnaces are easy to regulate, and thermal insulation is no problem. Larger furnaces are rarely "home made" in the laboratory.

Tungsten, tantalum and molybdenum (more or less converted into MoSi<sub>2</sub>) wires make excellent heating coils for specific applications. See R. Kieffer and F. Benesovsky [6] regarding compatibility of these metals with ceramic materials and insulating compounds at high temperatures.

Because of their low vapor pressure, Mo, W and Ta are well suited for building small high-vacuum furnaces. These furnaces are frequently operated under a glass bell. The available heating



Fig. 9. Stand for electrical furnace.



Fig. 10. Wooden core for making furnaces with internal heating coils. area is usually small, but very high temperatures can be reached. The heating elements of such furnaces are horizontally or vertically laid spirals of Mo, W or Ta wire. These must be well reinforced by a ceramic structure since these metals soften at high temperatures and are thus subject to plastic deformation. A heating element of this kind must be surrounded by metallic or ceramic radiation shields or by a cooled housing (similar to the apparatus shown in Fig. 12).

All apparatus parts are mounted on a horizontal base, drilled and fitted with vacuum-tight connections for cooling water and electric power. The base and the furnace are enclosed by a large glass or metal bell, making for a vacuum-tight assembly (Fig. 12) (K. B. Alberman; F. Davoine and R. Bernard [6]).



Fig. 11. Globar furnace: 1-External jacket (metal); 2-insulation layer (fire-clay grit, MgO, diatomaceous earth); 3-insulation support tube (fireclay); 4-Globars; 5-inner tube (Sillimanite, hard porcelain); 6-end plates drilled for the inner tube and the Globars.

#### GLOBAR FURNACES

Globar furnaces are much sturdier than most others but are less easy to regulate and, as they cannot be provided with as good thermal insulation, are also less economical. The furnaces are usually made with silicon carbide rods, though pipes are also in use. They may be used without major problems up to a temperature of  $1350^{\circ}$ C and for short periods, even to  $1500^{\circ}$ C. Good electrical contact at the conductor terminals is most important. Generally, Globars are manufactured with tightly wound adhering wire or metal ring connectors. It is not difficult to make a Globar furnace in the laboratory, as suitable supports, tubes and outer shields can be obtained ready-made.

It should be remembered that silicon carbide, a nonmetallic conductor, has a lower resistance when heated than when cold. Therefore, the furnace must be heated slowly, using a rheostat or a transformer and an ammeter. With rising temperature, the



Fig. 12. Tungsten furnace. 1-McLeod gauge; 2-brass base; 3-tungsten plate; 4-tungsten tube; 5-glass bell; 6-sample; 7-screw cover; 8-copper jacket; 9-copper cooling coil; 10-vacuum pump; 11-power supply; 12-cooling water input.

voltage should be lowered to avoid an undesirably high current. A fuse or a circuit breaker should be included in the circuit.

#### CARBON TUBE FURNACES

The heating element of these furnaces, which were first constructed by Nernst and Tammann, is a carbon tube. Because of their low resistance, they are also called short circuit furnaces. The larger models have found wide industrial use. Thus, these furnaces are commercially available. It does not pay to attempt construction of such a unit in the laboratory. The most expensive part is the transformer, needed because of the low resistance of the carbon tube, and this must be purchased in any case. Depending on the size, these units require some 100-1000 amp. at approximately 10 volts. Careful construction of the unit permits easy replacement of the carbon tube (whose durability at high temperature is limited). Temperatures of over 2000°C can easily be reached. A reducing atmosphere must always be maintained inside the tube. Should this be undesirable, then protective insert tubes must be provided. For this purpose, alumina can be used up to 2000°C. At higher temperatures, only sintered BeO or ThO, is effective; MgO is subject to reduction.

A variation of this type of furnace, with slotted graphite tube, has been described by W. J. Kroll [6] and has given satisfactory performance in various tests. Graphite is more resistant to oxidation than carbon. The disadvantage of its lower electrical resistance is overcome by dividing the tube into several current paths by appropriate longitudinal slots. This arrangement also functions in a vacuum or inert gas atmosphere.

#### TUBULAR TUNGSTEN FURNACES

Higher temperatures (up to  $3000^{\circ}$  C) are reached with tubular tungsten furnaces. Because tungsten is sensitive to  $O_2$ , and because of the improved thermal insulation, these furnaces must be operated in a vacuum or at least in an  $H_2$  or an inert gas atmosphere. A model with horizontal W tube is shown in Fig. 12. The tungsten tube, fixed in place with two sturdy, molybdenum-lined clamps, is supported by two heavy brass bus bars. The latter pass through a thick brass base plate, covered by a large glass bell.



Fig. 13. Tungsten furnace: htungsten tube; s-sight glass; st-radiation shields; v-vacuum bell; z-power input; the necessary vacuum connections to the base are not shown.

One bus bar is insulated from the base plate. The tungsten tube is surrounded by a copper sheet box, to which a tightly wound cooling coil is soldered. The brass base plate is drilled for two other tubes, which serve as connections for a McLeod gauge and a vacuum pump.

The connection to the vacuum pump must be of a large diameter. This furnace will reach 3000°C at 10 volts and appr. 1000 amp. and at a vacuum of  $10^{-5}$  to  $10^{-7}$  mm. In other models, the tungsten tube vertical (Fig. 13); alternais tively. a tubular sleeve made of tungsten plate may serve as the heating element (Fig. 14). (H. Bückler; R. Kieffer and F. Benesovsky [6].) Instead of tungsten, tantalum can also be used; the maximum permissible temperatures with tantalum are not quite as high as with tungsten (not over 2200°C). However, tantalum has the advantage of not becoming brittle through recrystallization, even after prolonged heating.



Fig. 14. High-vacuum furnace with a heating element made of tantalum plate (horizontal cross section). htantalum heating sleeve; *i*insulator; *s*-radiation shield made of Ta or Mo plate; *v*-vacuum bell.

An iridium furnace has been described by Von Wartenberg [6]. Since iridium has a considerable vapor pressure at high temperatures, the tube interior must be coated with a ceramic compound.

#### INDUCTION FURNACES

The energy of a high-frequency (for example, one megacycle) alternating current can be transferred through a large diameter coil to an electrical conductor, for instance, a metal or graphite crucible, which is placed inside the coil. The conductor is thereby heated. The ease of operation and the convenience of an induction furnace Thus, the red-hot crucible can be enare unsurpassed. closed in a cooled quartz tube, in which a high vacuum or an inert gas atmosphere can easily be maintained. However, the pressure range of  $10^{-2}$  to  $10^{1}$  mm. cannot be used because of the interfering glow discharge. With induction furnaces, temperatures of up to 3000°C can be reached very rapidly, in fact within seconds. Their disadvantage is the need for elaborate equipment, especially electrical apparatus, and the consequent high cost. Suitable current generators are commercially available. They are usually equipped with large transmitter tubes. It is best to make the furnace itself in the laboratory, designing it for the specific experimental purpose. Under special circumstances, a ceramic tube can be the energy receptor and thus serve as the heating element, provided the ceramic has a defect crystal lattice and consequently exhibits an intrinsic conductivity at high temperature (H. Davenport et al. [6]). Furnaces with the Nernst compound as a heating element and operated by direct current passage, which are sometimes recommended, have not proved to be satisfactory for normal chemical preparations.

#### ARC AND ELECTRONIC RADIATION FURNACES

Arc furnaces are useful in preparation of alloys and high-melting compounds with low volatility such as carbides, borides, lower oxides, etc. A small sample of the substance, a so-called button, is melted by a high-current arc under vacuum or in a suitable gas atmosphere at reduced pressure. The arc is struck between a suspended, cooled tungsten rod and a horizontal, cooled copper plate. The latter has cuplike depressions for melting the samples. Such furnaces are commercially available but can also be made without great difficulty in the laboratory (W. J. Kroll, G. Haegg and G. Kiessling [6]).

Several authors have described laboratory furnaces in which heat is transferred by electron bombardment (cathode rays). These are used for special applications [6].

Both furnace types have recently gained industrial importance for use with high-melting metals (Ti, Zr, Nb, Mo).

#### SOLAR FURNACES

Solar furnaces are suitable for special applications, e.g., for heating in a pure  $O_2$  atmosphere, in which other types of heating elements are corroded very rapidly. In a solar furnace, the sunlight is focused by a large parabolic mirror (e.g., 1.5 m. diameter). Very high temperatures are reached at the focus which, of course, must cover a reasonably large area [6].

#### Low Temperatures

Freezing mixtures or low-temperature bath (cryostats), cooled with solid  $CO_2$  or liquid nitrogen, are used for reaching temperatures below the ice point.

Ice is used for most freezing mixtures. Adequate crushing of the ice is important. This can be done in an ice mill or simply by pounding with a wooden mallet on an even concrete block  $40 \times 40$  cm., framed by a 10-cm.-high wooden strip. Such a block should be set up next to the refrigerator in any case, even if an ice mill is available. In this way the ice can be easily broken up before using the ice mill. The bad habit of throwing large chunks of ice into the mill and breaking them there with mallet blows will quickly ruin the most rugged ice mill.

temp.  $-21^{\circ}C$ 

temp.  $-40^{\circ}C$ 

temp.  $-56^{\circ}C$ 

temp.  $-27^{\circ}$ C to  $-30^{\circ}$ C

Freezing mixtures based on ice:

- 3 parts ice + 1 part NaCl
- 3 parts ice + 2 parts  $MgCl_2 \cdot 6H_2O$
- 2 parts ice + 3 parts CaCl<sub>2</sub> 6H<sub>2</sub>O
- 2 parts ice + 1 part conc. HNO,

The temperatures indicated for the last two mixtures can be reached only when the CaCl<sub>2</sub> or HNO<sub>3</sub> is precooled in a refrigerator. In all cases, ice and salts must not consist of coarse chunks but must be well crushed and properly homogenized. If ice is not available, mixtures of  $\rm NH_4NO_3$  and water (1 : 1; cools from +10° to -20°C) or KSCN and water (2 :1; cools from +10° to -25°C) can be used.

Lower temperatures may be obtained with solid  $CO_2$ , which can be purchased in blocks as "Dry Ice." If bought in blocks, it must be well crushed, preferably with a mill; or it can be produced as "snow" from a  $CO_2$  cylinder. To make "snow" a strong canvas bag is attached to the outlet valve of the cylinder. A short nozzle screwed onto the outlet is very practical. The cylinder is tilted downward and the valve opened as wide as possible. Strike the bag vigorously while the carbon dioxide flows into it (with a loud hissing noise) or the  $CO_2$  snow accumulating on the inner surfaces of the bag will clog the pores.

Even more practical than this primitive contraption is a hardwood box of about 0.75 liter capacity with a screw-on cover (Fig. 15). The cover has a groove around its circumference and is cut out to the diameter of the wooden box. A conical canvas bag is tightly fastened to this ring-shaped cover with a wire in the groove. A metal tube with a female adapter fitting the cylinder outlet is attached to the top of the canvas bag. The use of this small device is obvious. When the cover is unscrewed,  $CO_2$  snow can be easily removed from the box.

Solid  $CO_2$  in blocks can be kept in brassplated containers or in large Dewar flasks equipped with canvas bags, similar to those commercially available for food preservation with ice. Large glass flasks such as these Dewars are easily broken; thus, re-





moval and insertion of a bag or container requires the greatest caution.

Since solid  $CO_2$  is a poor heat transfer agent, it must be dispersed in a suitable liquid prior to use. Ether is not acceptable



Fig. 16. Device for removing liquid nitrogen from transport vessel. because of its high flammability. Acetone or the inexpensive methyl or ethyl alcohols are recommended. Trichloroethylene is especially suitable, because  $CO_2$  floats on its surface, thus preventing the mixture from foaming.

Liquid nitrogen is available nearly everywhere.\* Small liquefiers for laboratory use are also on the market. For transport, "safety cans" in various sizes are used. The liquid nitrogen is either decanted by means of a tilting mechanism or with a small siphon, which can be made by any glass blower (Fig. 16). A small rubber bulb provides the neces-

sary pressure for removing larger quantities.

Dewar vessels made of Pyrex are preferred. In the long run, they are much cheaper than thermos bottles because the latter break so easily. However, if one must use flasks of ordinary glass, the unavoidable breakage factor will be considerably reduced by prior rinsing with CCl<sub>4</sub> and slow cooling, while rotating the flask. One should never decant from large Dewar flasks and those made of ordinary glass. Such vessels should always be emptied by scooping out the contents. An appropriate scoop is made by soldering a brass cup (40 mm, wide and 60 mm, high) to a brass wire 40 cm. long and about 3 mm, thick, Smaller thermos bottles are emptied by putting a wet filter on their inner edge. It will freeze on immediately, and the contents can then be decanted. The liquid should always be poured out as quickly as possible. After the experiment is completed, the liquid nitrogen is poured back into the transport container via a sheet metal funnel. Glass or plastic funnels will generally crack.

\*Liquid air and liquid oxygen have been used in the past as laboratory coolants. This unsafe practice has by now disappeared almost completely in the U.S. Liquid air and liquid oxygen should never be used when a relatively inert coolant, such as liquid nitrogen, is equally suitable. This, of course, does not preclude laboratory use of liquid oxygen (for examples of the latter, see the section on Fluorine).

If one is forced to use liquid air in the presence of oxygensensitive compounds, the cooling flask should be covered with a protective jacket made of copper sheeting. The same protective measure should be taken when liquid air is used for cooling vessels containing activated charcoal (silica gel should preferably be used in these cases). Cold baths (acetone, methylene chloride, petroleum ether, pentane) are cooled by means of a copper coil soldered to a copper can (Fig. 17). Liquid nitrogen is forced with a siphon through one of the tubes into the can, where it evaporates and cools the bath fluid. The evaporated cold nitrogen gas escapes through the coil.

Constant temperature cooling baths, with temperatures ranging from -20 to  $-190^{\circ}$ C, can be obtained with liquid nitrogen slurries. The liquid nitrogen is mixed

with a liquid organic compound with suitable melting point, so that the latter partly solidifies. This slurry is capable of maintaining the melting point temperature for a considerable period of time. A few suitable materials are listed in Table 13.

Cooling blocks made of aluminum have many applications. These are machined as in Fig. 18 and provided with suitable wells for a thermometer and the vessel to be cooled. The block is suspended by a strong cord or in a gauze bag. The Dewar flask, filled with liquid nitrogen, is placed underneath the block, which then may be raised or lowered within the flask, depending on the temperature desired.

# **Constant Temperature**

No particular difficulties are experienced in controlling temperatures in the region from room temperature up to  $300^{\circ}$ C. Bimetallic strip devices, Wheatstone bridge circuits (ther-

mistor activated) or mercury thermometers with capacitance devices connected to relays can be used for control of bath temperatures. For good control, the immersion heater should have the lowest possible heat capacity and the bath should have as large a volume as possible. Should the bath volume be small for whatever reason, the heat capacity of the heater should also be low to prevent further bath temperature rise after the current is shut off. A thin Pt wire, wound around a frame, may be used as a heater in these cases; it may be placed directly in the bath without any insulation. Vigorous agitation of the liquid in the bath is important. Bath temperatures somewhat below room conditions may be maintained by means of an immersed copper coil with a constant



Fig. 17. Liquid N<sub>2</sub> cooling bath.



Fig. 18. Aluminum cooling block.

#### Table 13

Cold Baths o	f Low-Melting	Materials
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	M.p., °C
Isopentane (viscous when cold) Pentane Diethyl ether Carbon disulfide Acetone Chloroform Chlorobenzene Ethylene chloride Carbon tetrachloride	$\begin{array}{r} -158.6 \\ -130.8 \\ -116.3 \\ -112.0 \\ -95 \\ -95 \\ -63.5 \\ -45 \\ -25.3 \\ -23 \end{array}$

flow of cool water and a heater which regulates the temperature. If necessary, the water may be precooled by embedding a second coil in ice. Cryostats are preferred for temperatures below  $0^{\circ}$ C.

#### CRYOSTATS

At temperatures slightly below 0°C, the well-known Hoeppler thermostat is supplemented by a Dry Ice-filled vessel. However, lower temperatures are generally required and can be attained with liquid nitrogen. Various methods have been described in the literature; almost all of these are based on the principle of allowing liquid nitrogen to evaporate into a cooling coil placed in a cryostat. If the temperature becomes too low, a mechanically or electrically controlled valve interrupts the input of liquid nitrogen. One such cryostat [Peters, Chem. Fabr. 7, 47 (1943)] has a mechanical regulator actuated by the difference in expansion between an aluminum tube and a quartz rod inserted in it. The regulator is contained in an aluminum cooling block and operates a small valve at the outlet of the liquid nitrogen-carrying cooling coil; the latter is also fused to the Al block. The inlet of the cooling coil is connected to the bottom outlet of a liquid-nitrogen-filled special Dewar flask. If the temperature falls below the desired level, the valve closes. No additional liquid nitrogen can then enter the cooling coil because of the pressure in it. When the temperature rises, this pressure is vented through the opened valve and more liquid nitrogen enters. The storage vessel may be too small for lengthy experiments. It may be replaced by a 5-liter Dewar flask and the cooling block may be replaced by a copper cooling coil (Fig. 19), in which





case the fresh liquid nitrogen flow is controlled by a double-walled siphon. The connection between the copper coil and the glass should be inside the bath to reduce coolant usage. A Mariotte-type bubbler bottle provides the pressure head required. This self-explanatory setup, shown in Fig. 19, works very well. The rubber stopper of the liquid-nitrogen feed vessel can be provided with a sheet metal funnel to facilitate filling. The funnel is stoppered by a cork. A cork float, attached to a thin glass rod, permits easy observation of the liquid level from the outside. Naturally, a vessel with a window can also be used. A similar device, but with electrical controls, has been described by Zintl and Neumayr [Ber. dtsch. chem. Ges. 63, 234 (1930)]. It differs from the above-described apparatus in that the vapor pressure thermometer actuates an electrical control system. The essentials of this device are shown in Fig. 20. The downpipe f to the vaporizer d must be larger than the inner tube of the siphon h so the liquid nitrogen is transferred drop by drop and is not sucked in by the siphon. The tubes of the copper vaporizer must be connected with glass tubes within the bath, using short pieces of rubber tubing. Metal projecting outside the bath would lead to serious thermal losses.

Using this equipment, temperatures such as  $-50^{\circ}$ C can be maintained with a fluctuation of only  $0.05-0.10^{\circ}$ . To maintain a cold bath of 2.5 liters of acetone in an unsilvered Dewar flask at  $-50^{\circ}$ C for four hours, about one liter of nitrogen is required.

Thermostats filled with boiling liquids at constant pressure are also quite versatile. Temperatures lower than -196°C can sometimes be reached by evaporating liquid nitrogen at reduced pressure. The temperature is held constant by maintaining a specific pressure with a manostat. Details of these devices may be found in Grubitsch [1].


Fig. 20. Cryostat with electrical control: b) bath liquid; d) copper vaporizer; f) down pipe; h) siphon; k) condenser; l) liquid nitrogen; r) relay;  $s_1$ ) power supply of about 2 volts;  $s_2$ ) power supply of about 18 volts; t) vapor pressure thermometer; u) pressure head regulator (manostat); v) electromagnetic gas valve; the valve plunger must be sufficiently heavy not to stick in its seat.

# HIGH-TEMPERATURE THERMOSTATS

Thermostats can be controlled to  $300^{\circ}$ C by contact thermometers; expansion regulators are used up to  $500^{\circ}$ C. The latter are based on the difference in linear expansion between a quartz rod and an aluminum or iron tube. Alternatively, a contact thermometer may be located in a cooler part of the furnace, provided a fairly constant temperature difference can be maintained between the hot reaction area and the cooler measurement section. The frequently recommended arrangement consisting of an auxiliary furnace in series with the main furnace, whereby the contact thermometer is placed inside the smaller furnace, can only work if both furnaces are very well insulated and have equal heat losses. This is not easily achieved in practice.

Detailed description of the great variety of readily available commercial devices for high-temperature control is beyond the scope of this book. These devices are well described in the catalogs of laboratory supply houses and instrument manufacturers. The available instruments range from simple and inexpensive to highly sophisticated ones, designed to give very precise control. Their proper application depends on circumstances and must be left to the ingenuity and skill of the individual experimenter.

#### **Temperature Measurement**

#### LOW TEMPERATURES

Pentane and alcohol thermometers can be used only for gross measurements. They are too unreliable for accurate measurement, especially at low temperatures. The vapor pressure thermometer described by Stock is very exact and easy to make (Fig. 21). Since its dimensions are shown in the drawing, it is sufficient to describe the filling process. After washing with

cleaning solution and water and drying in an air stream, just enough pure Hg is added to bulb c to fill the manometer. After the gas trap q and the thermometer stem have been closed off by fusion, vacuum is applied at h and the whole apparatus, including the Hg, is thoroughly heated. If the manometer is filled with a gas which can be completely condensed at the temperature of liquid nitrogen, the filling gas, from an apparatus sealed on to the vacuum flask, can be condensed at a. About 1 ml. of liquid is thus obtained. Opening i is then sealed off. The filling gas is then allowed to warm up to room temperature and evaporate. The excess escapes through m, so that the flask will now contain a gas atmosphere at a pressure exceeding atmospheric by about 25 mm. The gas is recondensed at a with liquid nitrogen. Opening k is now sealed off and the Hg is poured from bulb c into the manometer arm. The thermometer is now ready for use. If the filling gas cannot be completely condensed with liquid nitrogen, it is necessary to transfer the Hg before admitting the gas. The thermometer should be provided with a millimeter scale and attached to a suitable stand. Substances used for filling are:



Fig. 21. Vapor pressure thermometer.

CS<sub>1</sub> (+ 25 to -10 °C). SO<sub>2</sub> (-25 to -40 °C), NH<sub>3</sub> (-30 to -77 °C). CO<sub>2</sub> (-75 to -100 °C), HCl (-85 to -111 °C), C<sub>2</sub>H<sub>8</sub> (-100 to -150 °C);

and for even lower temperatures methane and oxygen may be used.

#### LIQUID-FILLED THERMOMETERS

These are suitable for temperature measurement up to 600°C. Their accuracy is poor at temperatures below the melting point of Hg. Mercury-filled thermometers are subject to minor changes during initial use; therefore, good thermometers should be artificially aged. Aside from this, the calibration should be checked from time to time, at least at the freezing and boiling points of water (correct for barometric pressure!). Ice must be finely ground and should be well agitated during the measurement. The boiling point of sulfur may be used as a high-temperature calibration point. Considerable inaccuracy may be caused by "emergent stem." This uncertainty is best avoided by bringing the whole thermometer to the measured temperature. If this is not possible, corrections must be made. The correction is  $k = (\alpha - \beta) \cdot (t - t_{\alpha}) \cdot h$ . Here  $\alpha$  = expansion coefficient of Hg,  $\beta$  = expansion coefficient of glass, t = indicated temperature,  $t_0 =$  average temperature of the emergent stem, in degrees. The value of  $\alpha - \beta$  for common thermometer glass is approximately 0.00016. The main uncertainty present lies in the temperature  $t_0$ . A "differential thermometer" is most useful for its determination. The material of this thermometer should be as close to that of the thermometer in use as possible. It is placed alongside the main thermometer in such a way that its indicated temperature is close to the final measurement. The differential thermometer is also provided with second, smaller scale below the first. It is positioned in such a way that the meniscus of the main thermometer and the point on the auxiliary scale of the differential thermometer which corresponds to that temperature. coincide. It is now easy to make the necessary corrections from the indicated data. The error can amount to several degrees.

#### **RESISTANCE THERMOMETERS**

These can be used over a very wide range. Their principle of operation is based on the large temperature coefficient of electrical resistance of Pt and Ni (for example, the resistance of Pt changes by 0.4% per degree). These thermometers are among the most accurate temperature measuring instruments. It is not difficult to make a resistance thermometer in the laboratory, but the commercial instruments are preferable. The high-temperature type consists of a mica cross inserted in a thin-wall quartz tube. A fine double Pt filament is wound around the mica cross.

Recently, instruments for measuring low temperatures have appeared on the market. In these, the Pt wire is fused into a fine groove of a glass tube and coated with a thin glass film. Such instruments have very low thermal inertia. For the most accurate measurements the wire should be aged artificially by heating it at 100° above the intended use temperature until there is no further change in its resistance.

A Wheatstone bridge is used for resistance measurement. Details can be found in Kohlrausch [1].

### THERMOCOUPLES

Thermocouples are used for higher temperatures, up to  $1600^{\circ}$ C. Table 14 gives the usual wire combinations.

Couple	Usable range, °C	Thermo- electric output, +20°/ +100°C
Copper/Constantan Iron/Constantan Nickel-chromium/Constantan Nickel-chromium/nickel Platinum-rhodium/platinum	$\begin{array}{r} -200 \text{ to } + 400^{\circ} (+ 600^{\circ}) \\ -200 \text{ to } + 600^{\circ} (+ 900^{\circ}) \\ 20 \text{ to } + 700^{\circ} (+ 900^{\circ}) \\ 20 \text{ to } + 900^{\circ} (+ 1200^{\circ}) \\ 20 \text{ to } + 1300^{\circ} (+ 1600^{\circ}) \end{array}$	3. 45 mv. 4. 32 mv. 4. 96 mv. 3. 22 mv. 0. 54 mv.

Table 14

The figures in parentheses refer to permissible limits for intermittent use only; the other figures, to temperatures permissible for continuous use.

Please note: platinum thermocouples must not be heated in a hydrogen atmosphere to temperatures beyond  $900^{\circ}$ C in the presence of a Si-containing material (including quartz and ceramic substances), since their mechanical strength is greatly impaired by the uptake of impurities (K. W. Fröhlich, "On the stability of platinum at higher temperatures," Degussa-Metallberichte <u>1941</u>, No. 7). For special purposes requiring higher temperatures, thermocouples consisting of, for example, combinations of tungsten and molybdenum alloys may be used. However, because of various complications encountered, optical or radiation pyrometers are usually preferred in this temperature range.

The lead wires of these thermocouples are usually made of 0.35 to 0.5 mm. diameter wire. Sometimes it is undesirable to have thermocouple wires sufficiently long to bring the connections (which also form the cold junctions) directly to constant temperature. In this case the so-called compensators are used. These may be considered lead wire lengtheners. Such compensating wire may be ordered from the companies that supply thermocouples. The lead wires to the measuring instrument are then connected either directly to the thermocouples or to these compensators. The two wires are fastened together with screw clamps and the junction is kept at an exactly measured and carefully controlled constant temperature (0 or 20°C, glass capillaryina Dewar flask). Millivoltmeters equipped with a temperature scale are also supplied for most of the popular thermocouple combinations. The meter should have an internal resistance of a few hundred ohms, so as to make the lead wire resistance negligible. For more exact measurements the calibration should be checked from time to time against a few fixed reference points. Only when accuracy requirements are extremely high is it necessary to measure the thermocouple e.m.f. by means of compensation switches. Except when working with silicates (where the exposed junction may be immersed directly in the melt), the thermocouple is sheathed with a protective tube (quartz, unglazed procelain, mullite, alundum). The wires should be insulated from each other by thin quartz or ceramic tubes. Before use, the thermocouple should preferably be annealed for a short time by passing through it a sufficiently high electric current. Should the weld or the couple itself have been damaged, it can be rewelded with ease. The damaged part is removed and both ends of the wires are placed in a small, pointed natural gas-oxygen flame, and touched just when they begin to melt. A small bead of metal is formed and connects the two ends.

Should one of the lead wires have to be repaired, the resulting bead is carefully shaped with a hammer so that it will again fit into the insulation tube.\* For calibration of thermocouples, the following reference points are used:

Naphthalene b.p.	217.96°C	Gold m.p.	1063.	0°C
Tin m.p.	231.9 °C	Copper m.p.	1083	°C
Cadmium m.p.	320.9 °C	Lithium metasilicate		
Zinc m.p.	419.45°C	m.p.	1201	°C
Sulfur b.p.	444.60°C	Nickel m.p.	1453	°C
Antimony m.p.	630.5 °C	Palladium m.p.	1535	°C
Silver m.p.	960.8 °C	Platinum m.p.	1769.	9°C

\*The home-made and home-calibrated thermocouples described above should be used only if the commercially available products cannot be procured for one reason or another. Fabrication of a reliable and durable thermocouple, especially for use at high temperatures, is a delicate business and should preferably be left to the experts. Thermocouple manufacturers now supply literally hundreds of variations, with all kinds of shapes, lengths, diameters, protective sheaths and insulation. These thermocouples are also available in precalibrated form. Recently, special thermocouples with very fast response time and excellent stability, reproducibility

#### RADIATION PYROMETERS

For temperatures above  $1600^{\circ}$ C it is best to use radiation pyrometers, although these may certainly be used at temperatures above  $800^{\circ}$ C. Their operation is based on comparison of the intensity of the radiation emitted by the measured body with that of an appropriately chosen incandescent bulb. By adjustment of the (known) current fed to the bulb, the image of the filament, projected onto the image of the radiating object, is made to vanish. The temperature can then be read directly from the instrument scale.

Another instrument is based on focusing the total radiation emitted by the measured body on a blackened thermopile by means of a quartz lens. The e.m.f. of the thermopile then gives the temperature. Both the optical and the total radiation pyrometers are commercially available and are very convenient and easy to use.

# High Vacuum and Exclusion of Air

The chemists' requirements for high vacuum frequently differ considerably from those of physicists. For a physicist, high vacuum starts only when the mean free path of the gas molecules corresponds approximately to the dimensions of the container (somewhat below  $10^{-3}$  mm.), while chemists' requirements are much more modest. A chemist will frequently be satisfied with a good rotary oil pump. Among the many models commercially offered, those working with small, easily replaceable quantities of oil are preferred. Of these, the pumps operating on the "gas ballast" principle (that is, air is admitted during the compression cycle to prevent condensation) are very convenient, since they can also be used to remove easily condensable gases and vapors, without damaging the pump with condensate. These are perfectly satisfactory for simpler vacuum distillations, drying under vacuum, etc. When higher vacuum is needed they may be supplemented by jet ejectors or diffusion pumps. These last pumps require a forepump, since they work only at pressures ranging from 0.1-30 mm., depending on the type. The diffusion pumps are made of glass or steel and use Hg or a special oil (for example, silicone oil) for the vapor jet. Oil diffusion pumps have the great advantage of not diffusing Hg

and accuracy have been developed for missile and space use. These are usually metal-sheathed and the wires are insulated by compacted refractory powder. These sealed units, already provided with the necessary leads, are available in diameters as small as 0.5 mm. or less. Their cost is not exorbitant considering the many additional hours of stable operation gained by their use.

vapor into the vacuum, and thus do not need cooling devices for keeping the mercury out of the apparatus; however, they are more sensitive to reactive gases. Considering that in most cases the chemist working with high vacuum also uses Dry Ice or liquid nitrogen at the same time, he will derive no special advantage from oil diffusion pumps (even though those deliver a vacuum of less than  $10^{-5}$  mm, without cooling). Thus, most chemical laboratories use Hg pumps exclusively, except for special purposes. Since pump throughputs are usually rather modest, except for work involving vacuum furnaces, electric discharges, or molecular or thin film distillation, the usual pumps made of glass or steel, with a suction capacity of 1-5 liters/second, are perfectly satisfactory. Glass pumps are best heated with electric heaters, and a safety pan should be set underneath. Steel pumps of course obviate the danger of breakage. However, cleaning of steel pumps is not as simple as that of glass pumps, which require only rinsing with concentrated HNO<sub>3</sub>, followed by rinsing with water.

The most frequent mistake made in planning vacuum equipment consists in choosing tubing or stopcocks of too small diameter. The connecting tube between the pump and the apparatus should have an internal diameter of at least 15-20 mm.; stopcocks used on this line must have at least a 10-mm. bore. A simple calculation of the pumping capacity will show that even with lines of such diameter, a conduit length of a foot or so will reduce this capacity, at pressures below 10<sup>-2</sup> mm., by an order of magnitude or more! Therefore, the lines in vacuum equipment should be as short as possible, with the least possible number of stopcocks. The use of glass spirals, frequently recommended to make glass apparatus less rigid and more able to accommodate stresses, should be avoided as far as possible, since these spirals offer a high resistance to flow. To protect the pump in case of cooling water failure, a small, easily made device is used (Fig. 22). A small funnel with a small hole is pivoted and counterweighted. The cooling water passes through the funnel, keeping it constantly filled. Should the water flow fail, the funnel will be pulled up by



Fig. 22. Cooling water failure switch.

the counterweight as soon as it is empty, thereby closing the stopcock on the suction side of the pump (this may be either a pinchcock or an ordinary gas stopcock). Alternatively, the lever movement may actuate an electric tumbler switch which then breaks the circuit. Other devices based, for example, on a mercury manometer, may of course be easily designed. Electric flow switches to guard against interruption of cooling water are also commercially available.

#### PRESSURE MEASUREMENT

The first device is the ordinary U-tube manometer. Its two arms should be of equal diameter (not less than 10 mm.; for precision measurements, about 15-20 mm.) because of the meniscus depression. The calibration should preferably be etched directly on the tube and readings taken against a mirror mounted behind the tube so as to avoid parallax. In this way changes of 0.1 mm. may be estimated without difficulty. A cathetometer must be used for greater accuracy.

The construction of a good Hg manometer requires some care. The tube is first thoroughly washed with cleaning solution and distilled water and dried as described above under Cleaning of Glassware; then the filling unit is fused on (Fig. 23). The required quantity of carefully purified and distilled Hg is placed in the flask and constriction a is sealed off. The manometer is then evacuated with fore and diffusion pumps and the whole apparatus thoroughly heated by fanning with a flame, with the pump on. Following this,

the mercury is heated until it starts to boil in the vacuum, and constriction b is fused. The assembly is then allowed to cool. The mercury should not be allowed to distill into the tube while the latter is being heated and evacuated, as otherwise it will obstruct the U-tube and an air bubble will be left behind. The manometer is then tilted to pour the Hg into the tube, and the fused spot at b is carefully filed open. Never break it off, for the onrushing air stream will push the Hg so violently that the shock will break the manometer. Even narrowing the diameter at c, which is very useful, would not be able to prevent breakage if such a violent impact of the mercury against the glass were to occur. This is the simplest and most reliable method of filling a manometer; it is preferable to the often recommended distillation of mercury into the tube, which does not always guarantee perfect filling. Filling with subsequent degassing, which is often done, requires considerable experience and patience and is unreliable; it also frequently cracks the manometer tubes.



Fig. 23. Mercury manometer with filling device.

When working with gases at varying pressures (high vacuum to slightly above atmospheric pressure), a manometer of the type shown in Fig. 24 should be used. One side of this manometer is connected to the atmosphere via a mercury spray trap which



Fig. 24. Mercury manometer with pressure release valve. Dimensions in mm.

serves as a gas outlet and pressure release valve when the pressure in the apparatus is too high.

A very convenient addition is a barometer tube of the same diameter, mounted next to the manometer; its reading then furnishes the zero mark for the manometer reading (see p. 67, Fig. 41).

When working with reactive gases which can contaminate the Hg, it is best to use a "null manometer" rather than cover the Hg with a layer of paraffinoil,  $H_{2}SO_{4}$ , etc., as has frequently been Such null manometers recommended. consist of a simple U tube filled with paraffin oil, H<sub>2</sub>SO<sub>4</sub>, silicone oil or bromonaphthalene, both arms of which are connected on top by means of a stopcock (Fig. 25). For high temperatures, these U tubes can be filled with molten tin. The quartz spiral manometers of the Bodenstein type (Fig. 26) are highly recommended, for in their case only quartz is in contact with the gases. They are now commercially available completely assembled with microscope or mirror for reading and require only careful mounting in a vibration-free location. With good Bodenstein instruments, pressure dif-

ferences of less than 0.1 mm. may be read. For shipping, such instruments are usually filled with glycerol so as to protect the very sensitive spiral from damage. Despite their fragility, most of these instruments will withstand even a one-atmosphere pressure difference between the inside and the outside of the spiral. Thus, there is usually no need to worry, should such differences occur as the result of a leak. Bodenstein gauges can safely tolerate temperatures up to 500°C without a change of the zero point; if the temperature of the spiral goes higher, the constancy of the zero point is not assured, particularly if large pressure changes accompany the temperature rise. If large pressure changes are avoided, the zero point will remain almost unchanged, even at 700°C. To measure small pressure differences, inclined tube manometers filled with bromonaphthalene or silicone oil may also be used; Hg develops too much friction in inclined tube gauges, since very fine tubes must be used.



Fig. 25. Null manometer.

Fig. 26. Quartz-spiral manometer, Bodenstein type.

Figure 27 shows how to couple a null manometer to the system. In this case, a Bodenstein manometer connection is shown.

LOW PRESSURE MEASUREMENT

The pressure gauge designed by McLeod, usually called simply "the McLeod," has been in use for a long time for measurement of pressures down to  $10^{-7}$  mm.; however, it registers pressures reliably only in the case of noncondensable gases. The Hg used in the gauge should be carefully purified and dried (heating in vacuum). The McLeod gauge is rarely used for preparatory work in its original form. Should the need for such a manometer



Fig. 27. Connection of a null manometer (Bodenstein manometer).

arise, the reader may refer to the pertinent literature (Kohlrausch, Grubitsch, Lux [1]). If measurement only serves for orientation purposes, the more convenient "Moser manometer" (Fig. 28 a, b)



Fig. 28. Shortened McLeod gauges.

or a "Vakuscope" (Fig. 28 c) is preferred. Both of these versions of the McLeod gauge are commercially available. As far as other types of manometers are concerned, e.g., instruments based on gas friction, ionization, thermal conductivity,\* etc., the reader should refer to the pertinent literature. These instruments are rarely important in preparative work. For rapid orientation as to the order of magnitude of vacuum in an apparatus, one can use a small discharge tube with two aluminum electrodes placed about 10 cm. from each other. (Alternatively, two aluminum foil pieces wrapped at the same distance around a glass tube in the apparatus may be used.) A high-frequency vacuum leak tester (or a spark coil) is connected to the electrodes; its discharge gives a green fluorescence at 0.05 mm., which disappears completely at<0.01 mm. Thermoelectric vacuum gauges (range  $10^1$  to  $10^{-3}$  mm.) are also useful for many chemical purposes.

### LEAKS

Hunting for leaks in vacuum equipment may sometimes prove extremely time consuming. Leaks are usually caused by careless cementing, poorly lubricated stopcocks or ground joints, or poorly fused glass connections. A small high-frequency apparatus is indispensable for detecting such spots in glass equipment. The equipment is evacuated to about 0.1-1 mm. and the suspected leak

<sup>\*</sup>E. von Angerer, Technische Kunstgriffe bei physikalischen Untersuchungen [Industrial Techniques Applied to Physical Research], p. 165.

points are brushed with the leak tester electrode. The inside of the equipment will glow slightly. Wherever there is a leak, the discharge spark will follow it. The leak is thus easily discovered by the brightly glowing path of the current. However, thin fused spots should not be touched, since these may be broken by the discharge. Leaks in the glass must be resealed with a torch, or else sealed with a drop of sealing wax or picein. Leaks at the stopcocks are harder to find; thus all suspected stopcocks may have to be regreased as a preventive measure. If possible, one should try to limit the area of search by successive shutting of the stopcocks, if those are present between sections of the equipment. Larger leaks are easily detected by the noise made by the entering air, or by creating a positive pressure in the apparatus and painting the suspected spots with soap solution. Another recommended procedure consists in passing a CO<sub>2</sub>-releasing hose over the evacuated equipment while the latter is brushed with the highfrequency tester. The color of the discharge will change from reddish to white at the leak. Rubbing the equipment with a piece of cotton wool dipped in alcohol will also change the color of the discharge wherever the alcohol has directly touched a leaky spot and the vapor has thus entered the apparatus. This method will give results only if one makes sure that the CO<sub>2</sub> or the alcohol vapor which might have entered will come in the path of the discharge: therefore one should wait for a while before continuing with the testing of further suspected spots. Cemented metal-toglass joints frequently are leak sites, as are pores in cast metal parts. Occasionally, substances that release gases may simulate a leak.

Excellent but expensive devices for locating leaks are commercially available; they blow a halogen-containing gas (for example Freon or difluorodichloromethane) from the outside onto the suspect spot; when this gas enters the evacuated apparatus through the leak, it creates an ionic current in an attached ionization tube equipped with a Pt anode. This signal is amplified and triggers optical or acoustical devices.

#### STOPCOCKS

An important factor in the choice of stopcocks is a sufficiently large sealing surface. In vacuum equipment, three-way stopcocks are a constant source of trouble and should be replaced either with two-way stopcocks or three individual ones. Stopcocks with hollow plugs are usually preferred to those with solid ones because they are lighter. The most important types of stopcocks are shown in Fig. 29. "Schiff" stopcocks (Fig. 29 a, b) should be used wherever possible; they stay more reliably leakproof, since no channels connecting the tubes can form.



Fig. 29. Various stopcocks (a, b are Schiff type).



Fig. 30. Lubrication of stopcocks.

and brought to body temperature (30- $40^{\circ}$ C); a thin ring of grease is then applied around the middle of the upper and the lower halves of the plug by means of a wooden rod. In somewhat larger stopcocks both rings are then connected with a thin strip of grease on a line 90° from the bore (Fig. 30). The plug is then pushed into the slightly warmed seat so that the stopcock is "open," and turned back and forth with slight pressure. Never turn so far as to close it. Only when the grease has been evenly spread and the air bubbles have disappeared from the ground surfaces should the plug be turned all the way around. This is the only way to obtain lubrication free from streaks. Vacuum stopcocks should be moved gently and slowly, so as to keep the movement within the flow rate of the grease layer, and to prevent the "tearing" of the grease film; otherwise streaks and channels will form, resulting in unavoidable leaks. With some experience, imminent exceeding of this limit will be clearly felt by a somewhat increased resistance to turning. If streaks have formed, the stopcock should be carefully cleaned before applying fresh grease.

Before greasing, the stopcocks are carefully cleaned with benzene or ether

Pipe cleaners, which are thin, 10-cm.-long brushes obtainable from tobacconists, are very practical for cleaning small diameter stopcocks. A wad of cotton wrapped on a thin copper wire or a wooden stick may be substituted for the pipe cleaners.

When working with gases or vapors that attack stopcock grease, other greases (etherproof grease,  $P_2O_5$ ,  $H_2SO_4$ ) may be used (but only temporarily in vacuum equipment, as the stopcocks rarely stay tight for a sufficiently long time). Sometimes this may be improved by sealing the upper and the lower part of the plug with grease, and applying the other sealing agent ( $P_2O_5$ ) to the middle only. In general, it is best to use greaseless valves such as the diaphragm valves made of Cu, Ag or Pt, the Bodenstein glass valves, the Stock mercury valves and the "breakseal" valves.

# STOCK MERCURY VALVES

Among the various Hg valves described by Stock the most important ones are the float and the frit valves: the float valves afford rapid passage of gases, but they function reliably only if made exactly to the measurements given by Stock. The float should be made of solid glass and have just one, very carefully machined, ring-shaped ground zone (Fig. 31 a). Opening the stopcock f will



Fig. 31. Stock-type Hg valves. Stopcocks d may be omitted, to avoid contamination of the Hg by grease. In c, S is a fused-on glass rod, the point of which touches the glass frit from above and affords smoother downward flow of the Hg. Measurements in mm.

let the air into e, pushing up the Hg, which then lifts the floats. Evacuation of e will lower the level of Hg, thus releasing the floats, which will drop (if they stick, tap lightly against the glass) and open the way for the gas. The pressure difference between the two sides of the valve should preferably be low when opening. A tube attached to a vacuum source, and available at all times on the working table, is connected to the various valves when these have to be opened. The stopcock f is closed after each movement of the valve.

The frit valves may be opened even when there is a considerable pressure difference between the two sides of the valve. The frits, which are impermeable to Hg, thus replace the rising floats (Fig. 31 b). They have the disadvantage of offering considerable resistance to the gas flow. A modification of the frit valve, described by Wiberg (see also the original model with filter candle by A. Stock [1]), is also commercially available. Its functioning may be readily seen from Fig. 31 c. The gas should flow from the top to the bottom.

#### BODENSTEIN VALVES

These valves, which may be made either of glass or quartz and can be heated, certainly represent the neatest solution of the problem of greaseless valves, since only glass is in contact with the gases (Fig. 32). They consist of a capillary tube the opening of which matches perfectly with a carefully ground and polished sphere attached to a glass rod. A glass capsule, sealed to both the capillary and the rod, is sufficiently elastic to allow slight movement of the parts toward each other. A spring compresses both parts with a force of about 14 kg., thus closing the valve. The two working parts can be slightly pulled apart by a screw working against the spring; stop pins prevent breakage resulting from turning the screw too far. This valve may also be used to introduce liquids into a vacuum, provided no solid particles are suspended in the liquid. Even microscopically small particles pressed onto the ground surface are almost impossible to remove; they may also damage the ground surface (since the spring exerts a strong force) and thus create a leak in the valve. In this case even long rinsing with cleaning solution will be of no avail and the glass part must be replaced. Should the glass break, it can easily be replaced. Remove by melting the Wood's metal that holds the part in place; take the broken part out of the seat and replace with a new one in such a way that the glass rod with its fused-on



Fig. 32. Bodenstein valve.

glass bead is sealed in the screw and the enlargement on the capillary tube in the slot of the aluminum support. Immobilize both by slipping small pieces of asbestos paper into the remaining free space in the slot of the support. Then the whole unit is placed vertically with the screw down and freshly melted Wood's metal is poured into the preheated screw seat. After complete cooling (the screw should be at "open"), place the valve horizontally and pour Wood's metal into the slot containing the capillary feeder tube. Should the valve be used in a warm water bath, litharge-glycerol cement may be used instead of the Wood's metal.

A slightly modified, very rugged version of the Bodenstein valve has been developed by Kistiakowsky and described by Vaughn [7]. The shutoff surfaces of this valve are not glass-on-glass but glass-on-AgCl, and the elastic sections of the glass capsule are concave rather than convex.

# BREAK-SEAL VALVES

For preparatory work, the rather expensive Bodenstein valves may frequently be replaced by a combination of fusing of connections (closure) and break-seal valves (opening), if it is sufficient to open or close a connection only once.

Since one rarely succeeds in satisfactorily fusing large diameter glass tubes while these are under a vacuum, the tube spot to be fused later in the experiment should be slightly narrowed, thickening it at the same time by slightly compressing and then pulling it. This will make it possible to close the opening when the apparatus is under vacuum. To do this, the spot to be fused is simultaneously heated and pulled in the direction of the tube axis, or if this is impossible, it is pulled sideways with a glass rod.

Opening of a tube connection may be achieved by building in a break-seal valve (Figs. 33 and 34). This consists of a fairly large diameter glass tube with a fused-in smaller tube, whose tip has been pulled to a fine capillary bent into a hook. Alternatively, the inner tube end is blown out to a very thin-wall sphere. Before making such valves, one should practice making sufficiently thin-wall spheres or fine points, which can be reliably broken off later (however, it may be preferable to buy ready-made valves). Before enclosing the valve in the large tube, insert the "hammer,"



Fig. 33. Break-seal valves: h) magnetically operated hammer bar.

a piece of iron rod encased in a glass tube (for example, a piece of a large nail, fixed in the encasing tube with small pieces of asbestos wool before fusing). The valve is set vertically while the hammer bar is held by means of a strong electromagnet and set carefully on top of the break-off sphere or point. The valve is then ready to be fused to the tubes connecting with the other parts of the apparatus. When the valve is to be opened, the hammer is lifted a few centimeters with the electromagnet and dropped onto the point or the sphere, which will break, opening a path for the gas. The break-seal valve may also be set up horizontally, and the hammer appropriately directed by means of the magnet so as to shatter the point. In addition to this design, many other similar models have often been described, all based on shattering a capillary tube or sphere ([7], Figs. 35 and 36).



Fig. 34. Briscoe breakseal valve. The capillary point b is still open during sealing to side 1, but several scratches have been made at a. Then b is fused by means of a very small flame introduced through c. with air entering at d. Next the valve is sealed to the other side, the Hg-filled hammer bar is carefully introduced and d is sealed. A sharp movement applied from the outside will shatter the valve.

U-shaped capillary tubes may also be used to replace valves. Cooling with liquid nitrogen causes the formation of a small plug of frozen gas, which will obstruct the capillary. When low boiling substances, which do not solidify at liquid nitrogen temperatures, are used, a drop of Wood's metal is introduced; in this case a slight enlargement must be formed at both sides of the bend. Depending on whether the metal is made to solidify in the enlargement or in the capillary, the valve will be open or closed.

We can discuss here only a few of the numerous recommendations on how to introduce small quantities of gas into a vacuum, while retaining precise control over the flow. In many cases it is sufficient to file fine grooves in the plug of a glass stopcock, starting at the bore. In these cases, stopcocks with inclined bores (offset arms) should be

used as much as possible. The Bodenstein valve is also usable for fine control, although the adjustment is not exactly reproducible. A metal needle valve is the best means of control.

# NEEDLE VALVES

If the presence of a packing gland is not objectionable, the usual Le Rossignol valve can be used for fine control. Packless metal values. in which a ground joint takes over the function of the gland and which can be attached to glass equipment by means of metal ground joints, can be readily made by a machine shop. Such a value is shown in Fig. 37. The needle is attached to a screw stem  $\alpha$ . A groove is cut in the stem in order to permit gas access. The top end of the value stem is rectangular in cross section and fits into a slot in the ground section. The handle on the ground part permits rotation of the stem. If it is also desired to avoid stopcock grease for sealing, a tombac tube value or a diaphragm value, in which a tombac tube or a brass or copper





diaphragm provides the seal, is well adapted for this service (Fig. 38). Further details on valves can be found in Mönch [2].

### VACUUM APPARATUS

A vacuum apparatus consists of a pump section, vacuummeasuring devices, and a specialized part, the design of which depends on the problem at hand. A Toepler pump (Fig. 39) is installed in those cases when gases not condensed at the temperature of liquid nitrogen are to be removed from the system and measured or investigated. In that case, the pump section of the vacuum apparatus assembly takes the form shown in Fig. 40. It is useful to set up this section on a movable stand, and in such a fashion that the assembly is readily understandable. The gases aspirated by the Hg pump can be compressed into a gas burrette, which is connected to the Toepler pump at b. Automatic Toepler pumps have also been devised by Stock. The level vessel on the Toepler pump permits the gases to be placed under positive pressure. A condensation trap installed before a prevents Hg vapor

from reaching the apparatus and protects the pump from reactive gases or vapors. It is also advantageous to design the Toepler pump with an oblong cylindrical pump space instead of a spherical one, with the oblong section placed on a slant so that its shape and position is like the boiler of the Hg distillation apparatus in Fig. 5 [cf. E. Zintl and A. Harder, Z. phys. Chem. (B) <u>14</u>, 265 (1931); also F. Seel, Chem.-Ing.-Technik 27, 542 (1955)].



Fig. 36. Vacuum breaker, P. W. Schenk type:  $\alpha$ ) indentation in the tube wall; b) thin eccentric capillary tube which breaks on touching  $\alpha$  when the ground joint is turned.



Fig. 38. a) Tombac tube valve;
b) diaphragm valve.

# b) diapiragin varv

# Special Vacuum Systems

Vacuum systems are used primarily for work with very volatile or with airand moisture-sensitive materials. The apparatus developed by Stock for the investigation of boron and silicon hydrides is the prototype of a system which may be generally used when working with sensitive materials and, with modifications, also in many other related cases. Where greased stopcocks can be used, this system can naturally be simplified by installation of stopcocks inof Hg valves. However, since stead stopcocks must be regreased from time to time, Stock valves are preferable. If

substances corrosive to Hg are present, Bodenstein values or break-seal values and sealing-off points may be used instead of mercury values. Figure 41 shows the Stock "Universal System." Initially, the mixture of volatile substances to be studied is condensed in a gas trap by cooling the latter with liquid nitrogen. Such a trap consists of a U tube which is enlarged at the bend to form a cylindrical vessel with a capacity of 25-50 ml. (Fig. 43 b). Other condensation traps (Fig. 43 a-e) can also be used. The wash bottle type (Fig. 43 a, c, d) is especially recommended if it is connected properly; that is, the gases should not be brought in through the inner tube, which is usually very readily plugged. If the substance to be investigated is initially condensed in trap a(detail of the apparatus, Fig. 42), then the system is closed to the atmosphere and evacuated and finally the section between q and f is

Fig. 37. Metal

needle valve.

Fig. 39. Toepler pump.

check

valve

66



Fig. 40. Pump section containing a Toepler pump and gas burette. 1) forepump; 2) diffusion pump; 3) forevacuum flask (3-5 liters); 4) drying vessel (P<sub>2</sub>O<sub>5</sub>); 5) forevacuum manometer; 6) Moser manometer or McLeod gauge; 7) condensation trap for Hg; 8) two-way stopcock; 9) Toepler pump; 10) gas burette; 11) level vessel.

closed off. Then trap a is heated on a bath with slowly rising temperature until a vapor pressure of 30-50 mm. is registered at  $m_1$ . The vapor pressure and temperature (vapor pressure thermometer) are noted, the valves g and k are opened while r is closed, and l is cooled with liquid nitrogen. Some material is allowed to



Fig. 41. Stock's Universal System. a) forevacuum pump; b) diffusion pump; c) forevacuum reservoir; d) drying agent.



Fig. 42. Detail of the Stock Universal System.



Fig. 43. Condensation traps (gas traps).

condense in l; then k is closed, and the vapor pressure at  $m_1$  is allowed to adjust at the same bath temperature. If the pressure is still the same as before, another portion is distilled into l. This is continued as long as the vapor pressure in a remains unchanged. When it decreases, one can distill into u and t. Thus the substance is split into different fractions, depending on the vapor pressure, without any loss of material. If an error has been made, all of the substance can be recondensed in the initial trap, again without loss, and fractionation started again. The collected fractions can be split further, those belonging together can be combined, etc. The constancy of vapor pressure at constant temperature from beginning to end of the distillation is the criterion of purity. In order to determine the vapor pressure, the distillation must, of course, be discontinuous and sufficient time allowed for complete temperature equalization. It may be mentioned that the vessels should be lightly tapped during distillation-Stock specified an electromagnetic

vibrator—and one should not distill too fast. Stock also described gas receivers (Fig. 44) for collection of materials which are gaseous at room temperature in order to avoid having to keep these continuously in liquid nitrogen. An ingenious withdrawal valve g permits removable storage vessels to be filled at various points of the system. This valve consists of two glass frits immersed in mercury. The gas can pass through only when their surfaces are pressed together.

Low-temperature distillation columns have been devised for improving fractionation. These greatly shorten the process and in very many cases are absolutely necessary for fractionation of mixtures of substances with very close boiling points. Clusius and Wolf [Z. Naturforsch. 2, 495 (1947)] have de-



Fig. 44. Gas collection vessel.

scribed such an apparatus, which is shown in Fig. 45. This microcolumn is similar to those described by Clusius and Schanzer [Z. phys. Chem. (A) 192, 273 (1943)], but has a capacity of 6-10 ml.

Besides vapor pressure, melting point and molecular weight may be used for characterization of the fractions. Stock devised a very simple apparatus for determination of the melting point (Fig. 46). The thin glass rod with an iron core is first raised with an electromagnet, and enough substance is condensed in the lower half to fill it about halfway. It is thoroughly melted and solidified. Then the current to the magnet is shut off, leaving the point c of the glass rod resting on the surface of the solid substance. The rod is observed while the bulb k is slowly heated in a thermostatic bath. The temperature read at the point at which the tip of the rod begins to sink is the melting point. To render the movement of the rod readily visible, a red or black glass bead is fused at d. To determine the molecular weight, a preevacuated flask of known dimensions and weight is connected to the system via a stopcock and ground joints. It is filled at a measured gas pressure and weighed again after removal. The molecular weight is easily calculated from the pressure, volume and weight. The "Stock buoyancy weighing" method is more rapid. It is based on the buovancy of a quartz bulb in the studied gas, the pressure of which is simultaneously and exactly measured. The buoyancy of the quartz bulb is compensated electromagnetically [cf. E. Lehrer and E. Kuss, Z. phys. Chem. 163, 73 (1933)].

While the Stock system, sometimes modified for special purposes, is the standard apparatus for manipulation of readily volatile



Fig. 45. Low-temperature fractionation system:  $h_1 - h_5$ ) stopcocks (preferably vacuum stopcocks);  $m_1 - m_3$ )manometers; f) crude gas container; intermediate condensation in trap u; r) microcolumn, capacity 6-10 ml., with a heating coil s of Pt-Ir wire, vacuum jacket and level limit for the cooling bath; r is shown enlarged in relation to the rest of the apparatus. The volume between  $h_3$  and  $h_4$  is known and serves for control of the withdrawal rate. The fractionated gas is frozen out in d and transferred to receiver k by cooling t to a low temperature.



Fig. 46. Melting point apparatus.

substances, work with liquid or solid air-sensitive substances necessitates considerably closer attention to the specific problems at hand.

Moderate exclusion of atmospheric oxygen or moisture can be accomplished by working in a dry box or a dry bag, which are quite convenient. The dry box shown in Fig. 47 consists of a large gas-tight box with a well-closed opening for introduction of substances, tools and instruments. It is equipped with rubber gloved openings for the hands. Tube connections in the side walls allow the inner space to be filled or flushed with dry air or inert gas. An air lock can also be mounted on a side wall for introducing or removing the substances without disturbing the internal atmosphere. In order to observe the work, the box has a glass cover, or it can be completely built of clear, transparent plastic. These or similar boxes are sometimes commercially available; for example, they are also used for work with radioactive materials.

For many purposes one manages with the much simpler dry bags. A bag of thin elastic transparent plastic (e.g., polethylene),





Fig. 48. Dry bag.

open at one end, has a hole at the closed end. A glass tube with a stopcock is cemented into the hole and sealed in securely with rubberstopper gaskets (Fig. 48). The materials and instruments, e.g., a weighing bottle, spatula, etc., are placed in the bag through the open end, which is then closed by folding over a narrow strip several times and clamping it between two wooden strips. The inside of the bag is evacuated through the glass tube and filled to a low pressure with protective gas. The items inside are manipulated from the outside; the tools and vessels are grasped through the soft walls of the bag [W. P. Pickhardt, L. W. Safranski, and J. Mitchell, Analyt. Chem. 30, 1298 (1958); further references are also given there]. P. Ehrlich, H. J. Hein and H. Kühnl [Chem. Ztg. 81, 329 (1957)] describe a similar apparatus in which the bag is mounted on a solid base plate.

For recrystallization of air-sensitive preparations and, in general, for production of pure preparations, an apparatus devised by Ulich (Fig. 49) is used more advantageously [H. Ulich, Chem. Fabrik 4, 278 (1931); a similar apparatus but with standard ground joints is given by F.



Fig. 49. Apparatus for dissolving and recrystallizing (the tube attachments cand d can also be located in the center sections). Frierichs, Chem. Fabr. 4, 318 (1931)]. The apparatus, the construction of which is obvious from the figure, is filled with inert gas through tubes c and d, and flushed or evacuated, as neces-



Fig. 50a. System for precipitation, purification and isolation in the absence of air: H) condensation trap of the vacuum system; J) reaction flask; K) condenser; M) flask with  $P_2O_5$ ; O) filter tube with extension O'and glass frit O''; 1-7) ground joints; e-J) stopcocks; m'b) metal stopcock on a tombac tube sch.

Fig. 50b. Water flask for precipitation and decantation: Q) gas inlet tube; S) tube for connection with precipitation flask with ground joint 5' and parallel ground joint VV'; T) drain tube for water; R and U) pressure tubes with pinchcocks.

sary. The substance to be purified is first dissolved in flask b. By inverting the apparatus, so that flask a is on the bottom, the undissolved material can be filtered off on firt F. The dissolving or filtration can be carried out either hot or cold. The middle section of the apparatus can be heated by solvent vapors from the heated flasks a (or b) or by a heating mantle, whichever is handier. The pure crystals are separated in flask a by cooling or by evaporation and removal of the solvent. The crystals and mother liquor are then separated by again inverting the apparatus and filtering through the frit. Washing, filtering and vacuum drying (cold or hot) of the crystals collected on the filter plate can be easily effected via tubes c and d. Finally, the flask can be emptied in a stream of inert gas.





Fig. 51. Apparatus for precipitation in the absence of air. Arrangement for drying and transfer of the precipitates: W—transfer vessel with adapter X; Y—storage vessel with sealing constriction Z; 2-10—ground joints; g-n—stopcocks.

Fig. 50c. Apparatus for precipitation in the absence of air. A water flask for precipitation and washing is included.

A closed system for the precipitation, purification and isolation of highly sensitive substances [e.g., Fe (OH)<sub>2</sub>] from aqueous solution with complete exclusion of air was developed by Rihl and Fricke [Z. anorg. allg. Chem. <u>251</u>, 405 (1943)]. This system has proved itself in practice. The apparatus and use may be explained with reference to Figs. 50a, b, c and Fig. 51. First, looking at 55a, the system is filled with dry, O<sub>2</sub>-free N<sub>2</sub> through  $m_{\mathcal{H}}$  (tombac tube with metal stopcock and metal joint). Now h and f are closed and the system is evacuated through  $\boldsymbol{e}$ . Then g is closed, air is introduced through f, and h is removed from the apparatus at ground joint 2. Next, J is filled with N<sub>2</sub> through  $m_{\mathcal{H}}$ , and cap 7 removed under a N<sub>2</sub> stream. Then vessel J is filled with the solution (which passes through the fritted glass filter to retain the solid particles). The filling port is at 7. Using N<sub>2</sub> pressure, the solution is forced through the filter into  $J_1$  In the meantime distilled  $H_2O$  is boiled in flask P (Fig. 50b) in a stream of N<sub>2</sub>. After closing  $\hat{R}$ , the flask is inverted, and by brief openings of the pinchcocks at R and U, S and T are filled with pure, air-free water. Now the filtration sections N and O are removed and ground cap 5 is attached under a constant stream of N<sub>2</sub>. Thus, the apparatus reproduced in Fig. 50c is obtained. Since there is a steady stream of  $N_2$ , no air can enter. Now water is permitted to enter J, thus precipitating the solution in the flask. This is permitted to settle; tube S is pushed close to the surface of the precipitate by moving the parallel joint VV' and N<sub>2</sub> pressure is used to force the liquid outside the system via Rand T. The precipitate is washed several times in a similar manner with water from P. Finally the large trap H is put back in place and the system is evacuated. The large trap is then cooled in liquid nitrogen or Dry Ice and the apparatus is left to stand overnight. This removes most of the water, which condenses in H. Next day the substance is usually quite dry. It is finally dried by opening the stopcock to flask M, filled with  $P_2O_5$ . The apparatus is again filled with N<sub>2</sub>, and the supplementary section is added (Fig. 51, top). By inverting the apparatus, the powder is transferred to W. Nitrogen may be introduced at m, J removed, the vacuum stopcock X added, and Y filled after evacuation. Finally Y, containing the desired substance, can be separated from the system by sealing at constriction Z. I. and W. Noddack [Z. anorg. allg. Chem. 215, 134 (1933)] describe a similar apparatus. It is shown in Fig. 52. With it, material produced under  $N_2$  and contained in filter crucible b may



Fig. 52. Apparatus for filtration in the absence of air.

be washed with a solvent in an inert atmosphere and dried in vacuum. Wash liquid is added through a; the system is later evacuated and the  $P_2O_5$  vessel then connected.

An additional device—a dumbell apparatus for precipitation and filtration in the absence of air—is found under the preparation of thiocyanogen in Part II, section on Carbon.

Numerous specialized types of apparatus have been described for carrying out reactions between solid and gaseous components or materials which are volatile at high temperature. In all these devices the reactions proceed in the absence of air. Many such reactions are carried out in apparatus which contains as its basic element a boat in a tube that can be evacuated or filled with inert gas. Some operations can be carried out from outside the tube (heating, magnetization); others may require direct manipulation (introduction and removal of substance, crushing, etc.). This can be accomplished using long-handled tools manipulated through one opening in a stream of inert gas. Only one such opening should be present in the apparatus. Because of the positive pressure exerted by the gas stream, these reactions can be carried out inside the tube without danger of penetration by air.

For the numerous designs devised to meet these problems the reader is referred to the original literature [8].

Optical distortion of the interior by the curved glass walls can be troublesome; this can be alleviated by using flat glass ports which are cemented to short ground joints (Fig. 53), by which they are attached to the tube. The tubes

they are attached to the tube. The tubes shown in Fig. 54 can be generally used for storage and transfer of air-sensitive solids. The common principle on which these containers are based is that they can be evacuated and so opened that an inert gas flow protects the opening from air. Their design is based on the Schlenk tube, which was devised for such purposes.



Fig. 53. Observation port for reaction tube.

The problem of pulverizing an airsensitive substance produced in an inert

atmosphere or in vacuum, in order to transfer it either to Mark capillaries for x-ray photographs or elsewhere for further reactions,



Fig. 54. Vessels for storage and filling of air-sensitive solids (Schlenk tube):

a-a model used in the laboratories of Darmstadt and Freiburg Universities; b-model of O. Schmitz-Dumont [Z. anorg. allg. Chem. 248, 196 (1941)]. Section I is for storage; Section II for transfer to glass containers or Mark capillaries. was solved by Zintl and Morawietz with a vacuum ball mill. It consists of a hollow bulb (10 cm. in diameter, 4 mm. wall thickness) (Fig. 55) containing grinding balls (10-15 mm. diameter). At the left of  $S_2$  the tube is constricted somewhat on one side so that it will not be closed off by a ball during removal of the milled material. The substance and the balls are introduced through  $S_1$  while the mill is purged with an inert gas. Then the mill is evacuated and rotated in a suitable device. Naturally one can also mill in an inert atmosphere, but working in vacuum is recommended, since loosening of the joints is thus prevented and, in addition, the fine powder is not elutriated as much. The speed of rotation is 70-80 r.p.m.; the milling time, usually 1-2 hours [Z. anorg. allg. Chem. 236, 372 (1938) (in which an arrangement for filling Mark capillaries is also described)].



Fig. 55. Vacuum ball mill.

Pulverizing large, compact pieces of solid material or material solidified in a boat or crucible can be accomplished by use of a small dental grinding apparatus in an inert atmosphere. For these methods, developed chiefly for air-sensitive alloys, see Part III, 5, section on Alloys and Intermetallic Compounds.

# "FARADAY SYSTEM" (TEMPERATURE GRADIENTS)

In some reactions between a slowly reacting metal and a volatile reagent, conducted in the absence of air, the metal must be heated as high as possible, but the pressure of the gaseous reactant must not reach a point at which it will burst the reaction vessel. For such experiments W. Biltz et al. have successfully introduced the "Faraday System" [E. F. Strotzer and W. Biltz, Z. anorg. allg. Chem. 238, 69 (1938)]. The apparatus consists simply of a sealed Vycor or quartz tube, with the metal and a supply of the other reactant (for example, S or P) at opposite ends of the tube. The "metal side" can then be heated to 700-1000°C, depending on the tube material, while the other half of the tube is held at 400-500°C, depending on the vapor pressure of the other reactant, until the reaction with the metal is completed to the desired extent. By subsequently lowering the temperature, the excess of the volatile component can be removed from the reaction product. By breaking the tube, if necessary in an inert atmosphere, the reaction products can be removed. The extent of the reaction can then be determined by subsequent weight determinations.

Similar devices with sealed tubes, which are placed in a definite temperature gradient, have proved themselves in synthesis and decomposition of metal halides (cf. H. Schafer et al., Z. anorg. allg. Chem. 1952 and subsequently).

#### Gases

As is evident from the preceding, inert gases are frequently necessary for changing from vacuum to atmospheric pressure. Even though gas production and purification are described in a special section (Part II), a few general points will be discussed here. Inert gases, N<sub>2</sub>, CO<sub>2</sub>, H<sub>2</sub>, O<sub>2</sub>, Cl<sub>2</sub>, SO<sub>2</sub>, NH<sub>3</sub> and many others are stored in steel cylinders. The valves of the steel cylinders are supplied with different threads: left hand threads for flammable gases, right hand threads for all others. Furthermore, the cylinder threads are not even always the same for all nonflammable gases. Thus, different valve designs may have to be used in each case. Usually, the cylinders are painted in code colors identifying particular gases. Confusion is minimized through these precautions.

Regulating valves are attached to the cylinder valves to provide flow control during delivery. One either uses simply a fine regulating valve, or a pressure reducing valve, whose spring-operated mechanism on the outlet side permits adjustment to a definite pressure, which is largely independent of the internal pressure in the cylinder. Such pressure-reducing valves are widely available. They sometimes are not able to assure a steady gas flow at the relatively low rates necessary in the laboratory. The flow itself is controlled by the needle valve at the outlet, after a pressure of about 0.5 atm. is established on the low-pressure side of the regulator, with the cylinder valve fully opened.

# GAS GENERATING APPARATUS

The most widespread apparatus is still the Kipp generator. For production of air-free gas, e.g.,  $CO_2$  for driving gases into nitrometers, it can be provided with an attachment that keeps the upper chamber completely filled with the gas and prevents the penetration of air into the decomposition acid. The operation of this apparatus is shown in Fig. 56. Honisch's variant of the Kipp generator, which assures complete utilization of the acid and prevents mixing of the used with the fresh acid, is increasingly popular, although the somewhat complex stopcock construction is sometimes troublesome.

A gas generator equipped with standard ground glass joints, which permits complete utilization of the acid and easy replenishment. has recently been made commercially available.

All these devices have the disadvantage of not permitting efficient degassing of the substances used to generate the gas. The apparatus shown in Fig. 57 is a more effective producer of maximum purity gases. Using this apparatus, gases can be generated from a liquid and a solid, or from two liquids. A liquid is always placed in the upper bulb. The entire apparatus can be evacuated before opening stopcock h and, if necessary, the liquids can also be degassed by boiling. A modification of the apparatus which uses, instead of stopcock h, a ground-in stopper s operated by a glass rod, has been devised by Bodenstein (Fig. 58). If the stopper is grooved down to the midpoint and the seat has a vertical groove below the midpoint, the introduction of the liquid may be controlled simply by rotating the stopper instead of raising it. If degassing of the upper liquid by evacuation is desired, the glass rod of the stopper must obviously be sealed against the atmosphere by a rubber stopper.







Fig. 56. Kippgenerator for producing air-free gases.

air-freegases.

Fig. 57. Kippgene- Fig. 58. Gasgenerarator for producing tor with Bodenstein dropping funnel. May be evacuated.

Special devices are used for the generation of a gas from two mixed liquids (see the preparation of HCl gas from concentrated hydrochloric acid and concentrated H<sub>2</sub>SO<sub>4</sub>, Part II, Fig. 143).

# PURIFICATION OF GASES

The scrubbing action of an ordinary gas washing bottle is limited. In general, if the scrubbing liquid and the gas passed through it are to interact in a satisfactory manner, the flow rate

should not exceed 10 liters/hour. The flow rate can be roughly estimated from the frequency of the bubbles. With tube diameters of approximately 5-6 mm., one bubble per second corresponds approximately to a flow rate of one liter per hour in a simple wash bottle. A frequency of ten bubbles per second (i.e., that rate at which single bubbles can still be counted exactly with the naked eye) represents the upper limit for satisfactory interaction with the wash liquid. Wash bottles with fritted disks are considerably more efficient.



#### FRITTED-DISK WASH BOTTLES

The effectiveness of wash bottles can be greatly increased by good dispersion and subdivision of the bubbles. Such fine dispersion of gas bubbles is effected in fritted-disk wash bottles by means of a sintered glass disk through which the gas enters into the wash liquid. For especially efficient scrubbing, a fritted glass disk of medium or coarse porosity can also be inserted at the bottom of a long glass tube, which holds the wash liquid. The disk is held in place with a rubber stopper or by fusing to the tube. In a simpler device, such as shown in Fig. 59b, a plain fritted glass filter f is attached to a glass tube g of the same diameter.

Especially fine bubble dispersion can be attained with an unglazed clay cell used as a frit. The cell is attached to a glass tube with a rubber stopper, and the assembly is held at the bottom of a large-diameter glass tube by means of a second rubber stopper (Fig. 59a). The rubber stoppers are secured in place with wire or clamps. Finally, the scrubbing effect can also be increased by filling the wash vessel with glass fragments or glass beads.

For runs of longer duration, in which consumption of the wash liquid must be taken into account, a wash tower may be used. It is similar in design to that shown in Fig. 59b, but has a dropping funnel on top instead of the stopper, and a stopcock for removal of spent wash liquid is fused on approximately at the level of g. It is then possible to add fresh wash liquid during the operation and to drain off a corresponding amount below.

# DRYING OF GASES

Drying is carried out either with concentrated  $H_2SO_4$  in wash bottles, or by passing the gases over a drying agent in drying tubes, or by cooling to low temperatures.

Drying agent	mg. H <sub>2</sub> O/liter	Dew point, °C	Remarks
$     ZnCl_2     CaCl_2 $	0.8 0.2	-33	Not to be used with $NH_3$ , amines, HF or alcohol.
CaSO <sub>4</sub>	0.005	-63	Neutral. One of popular U.S. brands is Drierite.
СаО КОН	0.003 0.002	67 70	Not to be used with alkali-sensitive gases and those con- taining CO <sub>2</sub> .
Conc. H <sub>2</sub> SO <sub>4</sub>	0.003	<u> </u>	Not to be used with H <sub>2</sub> S, NH <sub>3</sub> , HBr, C <sub>2</sub> H <sub>2</sub> or HCN.
Silica gel	0.002	-70	Can be regenerated at 200-300°C.
$Al_2O_3$	0.0008	75	Very effective.
$Mg(ClO_4)_2$	0.0005	-78	Caution when used with organic substances!
$P_2O_5$	0.00002	<b>—</b> 96	Not to be used with NH3 or hydrogen halides.

Table 15

Drying	Agents	$\mathbf{for}$	Gases
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The drying agents listed before silica gel are virtually unusable in high vacuum.

The action of a drying agent is obviously limited. The degree of drying that can be attained with any drying agent depends to a great extent on the experimental conditions. The equilibrium state, i.e., the humidity corresponding to the partial pressure of water vapor in the drying agent, is practically never attained. For this reason, the data of different investigators on the degree of drying that can be attained with individual drying agents differ greatly. The values shown in Table 15 give a survey of the approximate effectiveness of the most commonly used drying agents.

In particular, the following drying agents should be mentioned: CaCl<sub>2</sub>. The values in the table are valid only for the fresh, anhydrous salt. The drying efficiency can decrease greatly, especially at higher ambient temperature and on prolonged use.

Silica gel is available commercially as a blue gel containing a color indicator. With or without indicator, it is sold as gel pellets approximately 3 mm. in diameter. The indicator form offers many advantages. On exhaustion, a sudden color change from blue to light pink takes place. At the color change, the partial pressure of  $H_2O$  corresponds to approximately 1.5 mm.

It can be regenerated readily and as often as desired (drying oven at  $200-250^{\circ}$ C), but must be used with caution, since, like the even more active Al<sub>2</sub>O<sub>3</sub>, it can also absorb large quantities of many gases other than H<sub>2</sub>O, and this can be the unsuspected cause of many an unexplained experimental loss of material.

Phosphorus pentoxide generally contains traces of lower oxides of phosphorus, which can be troublesome, since they form PH<sub>3</sub> with  $H_2O_2$ . It can be tested by heating its solution with AgNO<sub>3</sub> or with HgCl<sub>2</sub>; reduction occurs if lower oxides are present. More rigorous work requires  $P_2 O_5$  that has been sublimed in a current of  $O_2$  before use, preferably directly into the drying tube. For more reliable oxidation of all impurities, the sublimation may be followed by short contact with a Pt catalyst. The drying tube is connected to an equally long and, if necessary, somewhat larger diameter glass tube. The connector is a small glass tube approximately 5 mm. in diameter and 6 cm. long. It is fused to the two larger tubes. The drying tube is filled very loosely with glass wool and dried at a high temperature in a stream of dry air (a CaCl<sub>2</sub> tube is sufficient). Some Pt foil is now placed in the empty tube and, following that, sufficient  $P_2O_5$  is introduced in such a way that a free channel for gas passage is available over the entire length of the tube. A slow stream of  $0_2$  is now passed through. The tube with the glass wool is heated somewhat and then the area of the Pt catalyst is heated almost to the softening point of the glass. By heating to 200-300°C, the  $P_2O_5$ slowly sublimes and passes over the catalyst. The heating should start at the Pt foil and proceed gradually to the other end. The  $P_2O_5$  is uniformly distributed in the heated tube containing the glass wool (Fig. 60).

Smaller drying tubes are also filled with glass beads or small pieces of glass tubing. Figure 61 shows a usable form which, because of the vertical position of the ground joints, can be readily interchanged. Loose magnesium perchlorate (without a carrier) as well as silicagel can be readily poured in, since these materials do not plug the tubes as readily as  $P_2 O_5$ .



Fig. 60. Purification of phosphorus pentoxide by sublimation.

The neatest and most efficient solution of the problem of the drying of gases is cooling to low temperatures. In this procedure

the gas is passed through a spiral condenser immersed in liquid nitrogen or in a Dry Ice-acetone mixture.

NOBLE GASES

While discussing purification operations, we should consider the purification of noble gases more specifically, since these are not treated in Part II. Of these gases, argon is an unexcelled protective gas for highly sensitive substances. Helium and neon are equally effective and are also used occasionally. These gases are stored in steel cylinders and can be freed of the usual

impurities  $(H_2O, O_2, CO_2)$  by the methods customarily used for other gases (e.g., N<sub>2</sub>). A simple and very effective method for the removal of moisture and O<sub>2</sub> from inert gases and from N<sub>2</sub> or H<sub>2</sub> is given by Harrison [9]. In this method the gas is passed through a U tube (e.g., 40 cm. high, 2.5 cm. in diameter) partly filled with liquid Na-K (25% Na, 75% K).

In addition, the problem of the removal of small amounts of  $N_2$  from the inert gases, particularly from Ar, requires special consideration here. The  $N_2$  may be combined with metallic Ca, Mg, Ti or Zr; metallic U is also very useful.

Because of the low reaction rate, absorption on Ca, which is thus converted to  $Ca_3N_2$ , must be carried out at 600-700°C, and requires an iron tube to hold the Ca turnings. The apparatus is shown in Fig. 62. The tube is approximately 70 cm. long and made of stainless steel. In addition, the Ca should be activated by a small amount of  $Ca_3N_2$  to speed up the reaction (O. Ruff et al. [9]).



Fig. 61. Phosphorus pentoxide tube.

Similarly, Mg turnings can be activated with metallic Na; they quickly absorb N<sub>2</sub> at approximately  $600^{\circ}$ C. Several small pieces of Na are added to the Mg turnings before the first heating of the tube; the Na distributes itself automatically (Grube and Schlecht [9]).



According to N. W. Mallet [9], N<sub>2</sub> can also be removed from inert gases by means of Ti, Zr or U. As shown in Fig. 63, a quartz tube, approximately 2.5 cm. in diameter and approximately 70 cm. long, is used. The Ar, predried over  $Mg(ClO_4)_2$ , is passed through it. The Ti powder is placed between two steel wool wads in a layer 25 cm. long, containing approximately 150 g. of metal. The particle size of the Ti powder should not be too low. The metal should be loosened up by pulverizing after use, the frequency of such treatment depending on how it is used. The sintered material should be regenerated and the entire tube packing should be treated after it has been in operation for a total of eight hours. Because of sintering. a working temperature above 850°C is possible only if Ti is present in the form of coarse turnings. Before it is used, ordinary commercial Ti must be freed of its generally high  $H_2$  content by heating in vacuum. When 30-40 ml. of  $N_z$  (or  $O_z$ ) per gram of Ti has been absorbed during a run, the absorptivity of the packing is greatly decreased and the material must be regenerated. A packing can be thoroughly exhausted, without any danger of Nabreakthrough, if a similar second tube is installed after the first.

A more recent method for purification of inert gases, which can also be used for  $H_2$ , consists in passing the gases through liquid




magnesium at  $750^{\circ}$ C. A steel apparatus is used, such as that shown in Fig. 64 (I. Jenkins and D. A. Robins, Third Plansee Seminar, Reutte 1958, and private communication).

### FLOW RATE

Flow rate can be estimated through bubble counting. It is measured with a rotameter or a differential monometer. The rotameter is



Fig. 64. Purification of inert gases with liquid magnesium. based on the displacement of a rotating float in a slightly tapered, calibrated tube. Both the tubes and the floats already properly calibrated, are commercially available.

In differential manometers, the two arms of a U tube are bridged by a calibrated capillary tube through which the gas flows (Fig. 65). Flow rate is proportional to the pressure difference between the two sides of the capillary.

For exact measurements, the temperature of the capillary should be constant. A glass stopcock is connected parallel to the capillary tube. Overflow of the manometer liquid can be prevented by opening

the stopcock at the last minute; experience shows that attempts to prevent trouble by controlling the flow rate usually come too late. Addition of a filling tube to the manometer permits pouring in the liquid after installation and drying of the manometer. The lip is then sealed off. Suitable filling liquids include: concentrated  $H_2SO_4$ , paraffin oil, silicone oil, bromonaphthalene, Hg or even stained water, depending on the purpose. By using inclined tube manometers, even very small flow rates can be measured quite well, employing only slightly constricted capillary tubes. Such devices are commercially available and can also be prepared in the laboratory.

The capillary tube shown in Fig. 65b can be readily interchanged. A calibration is required for each gas.

#### VOLUME MEASUREMENT

Calibrated gasometers or gas burettes are used. A constantpressure gas burette was described by Schenk [Z. anorg. allg. Chem. 233, 393 (1937)]. Otherwise, wet test meters—commercially available in precision types—are used. They are especially useful in calibration of flow meters.

### GAS RECEIVERS AND STORAGE VESSELS

Gases are stored in gasometers. The chief disadvantage of the simpel gasometers is that, with the exception of bell-type devices,



Fig. 65. Differential manometer.



Fig. 66. Constant pressure gasometer.

they do not permit removal of the contents at a constant pressure. More recent models avoid this difficulty by having the feed line from the upper container connect to an overflow flask instead of reaching the bottom of the reservoir. Such gasometers can be easily improvised from large flat-bottom jars or even 60-liter flasks. The arrangement of the tubes can be seen in Fig. 66. Tube  $d_{\star}$ extending to the bottom of the bottle, serves for piping the sealing fluid back to a. This can be effected either by putting the lower container under positive pressure or by creating a vacuum in the upper one. An immersion tube f in the upper container makes this vessel a Mariotte flask. The gas pressure is determined by the difference in height g between the outlet of f and the overflow tube in the lower container. Such gasometers function very dependably. Once set, the flow rate remains completely constant for many hours. The one disadvantage is the necessity for large quantities of sealing liquid. The use of saturated, degassed common salt solution as sealing liquid reduces the danger of carry-over of impurities. For some purposes suitable paraffin oil may also be used. It must be heated and outgassed invacuum prior to use. Bell gasometers in which the bell floats in a narrow, ring-shaped slot filled with Hg offer even better protection against impurities. Because of the considerable quantities of Hg, which they still require in spite of their special construction, they are limited in size. As can be seen in Fig. 67, a wooden shield serves to guide the bell. When a rigid connection between gasometer and apparatus is desired, the gas can be removed via a glass tube fused at f (dashed lines, Fig. 67) instead of through h.

Storage is often simplified if the gas is forced into evacuated steel cylinders. A small pressure vessel, cooled with liquid nitrogen, is used for this purpose. The gas is liquefied in this container and then flows through a copper capillary into the evacuated steel cylinder (Fig. 68).



Fig. 67. Bell gasometer.

Fig. 68. Transfer of condensable gases into steel cylinders.

If it is desired to prepare a gas mixture in the steel cylinder (e.g., by filling it first with  $N_2$  to 30 atm. and then adding  $H_2$  until the pressure totals 120 atm., thus obtaining  $NH_3$  synthesis gas) one should remember that gases under such high pressures have a considerable viscosity, which hinders complete mixing for days. Therefore, convective mixing should be induced through heating. This can be done by placing the cylinder in an inclined position, with the valve on the bottom, and heating it by irradiation with a 60-watt light bulb. Occasional analysis of the mixture serves as a useful check of the constancy of gas composition.

# Liquefied Gases As Solvent Media

A great deal of equipment has been devised for work with liquefied gases, such as  $NH_3$ ,  $SO_2$ , HF and others. This equipment permits carrying out such operations as precipitation, filtration, washing, drying, titration, etc., in complete absence of air and moisture. The operations proceed either in vacuum at the vapor pressure of the particular liquefied gas or in an inert atmosphere. The following brief description of equipment cannot make any claim to completeness.



Fig. 69. Apparatus for reactions in liquid ammonia: I) reaction tube; II) condensation equipment; 2 and 6) pressure release valves; 3) condensation, drying and storage of  $NH_3$ ; 4) protection against spray and fog-fritted glass or glass wool.

Zintl and Kohn [10] describe an apparatus for reaction of a salt with a solution of alkali metal in liquid NH<sub>3</sub>. This apparatus may be adapted to other similar uses (Fig. 69). The basic component is an H-shaped tube. Leg b contains the salt (this leg is later sealed off at c); leg a contains the alkali metal. A flexible connection (e.g., a tombac tube) to the other parts of the apparatus is provided at 5. After evacuation, the NH<sub>3</sub> is introduced through 1 and condensed in 3, which contains sodium for drying the NH<sub>3</sub>. From 3, the NH<sub>3</sub> is distilled into a and b and condensed there. Franklin and Kraus [10] showed that NH<sub>3</sub> distilled once over Na is completely clean and dry. When the Na in a has dissolved, the H tube is tipped and the solution is poured through e into b. The connecting tube fserves for pressure equalization. At the end of the reaction the NH<sub>3</sub> is evaporated, the apparatus evacuated, and b cooled in liquid nitrogen to loosen the substance from the walls. The apparatus is then filled with  $N_2$  and cut open at c, and the substance is crushed with a stirring rod inserted through c while a stream of N<sub>2</sub> passes through. For washing, the substance is transferred in a stream of N<sub>2</sub> into a well-dried "washing tube" (Fig. 70) (A. Stock and B. Hoffman [10]), which contains a small inset tube surrounded by glass wool or small glass beads. After filling, the N<sub>2</sub> inlet tube is sealed off. The tube is evacuated, NH3 is introduced and condensed, and then the top of the "washing tube" is sealed off. The tube is allowed to warm to room temperature and cooled on top with water or ice. The NH3 refluxes condenses on top, flows down and extracts the substance on the glass wool. The extract runs down through the wool while the fresh  $\rm NH_3$  distills upward through the small inset tube.





Fig. 70. Wash tube for extraction with liquid ammonia.

Fig. 71. Extraction with liquid ammonia.

To separate a mixture of solid substances with different solubilities in liquid NH<sub>3</sub> (or SO<sub>2</sub>) by extraction, the apparatus pictured in Fig. 71 is used. It was devised by Biltz and Rahlfs and improved by G. Jander, Wendt and Hecht [10], as well as by Klement and Benek [10]. The solid starting material is placed in leg a on the fritted glass disk b. Purified NH<sub>3</sub> gas is then admitted through c and condensed in a by cooling. After the portion soluble in NH<sub>3</sub> has dissolved, the cooling bath is transferred to leg d and the NH<sub>3</sub> is forced by its own vapor pressure from a into d. The undissolved residue remains on filter b in a. Fresh NH<sub>3</sub> is condensed in a by cooling a instead of d and opening the connecting stopcock. By repeating the operation, the insoluble residue can be multiply extracted with NH<sub>3</sub>. The NH<sub>3</sub> can be either directly recycled to aor removed as a gas through c, condensed outside the apparatus, and recycled. By using the special head q, equipped with a cooling jacket holding Dry Ice-acetone mixture, instead of the ground glass cap f, the extraction can be carried out with NH<sub>3</sub> continuously trickling from q to a. The extract collects in d and can be removed through e.

No foreign gases are permitted in the free space of all such apparatus or the recondensation of  $NH_3$  (or  $SO_2$ ) will be appreciably hindered. Thus, the free space is evacuated from time to time,

while the liquid  $NH_3$  (or  $SO_2$ ) is held back by cooling one leg. The temperature should not be lowered to the point where the liquid freezes, because of the risk that the filter plate may shatter when rewetted.



Fig. 72. Apparatus for reactions in liquid ammonia.



Fig. 73. Flask for reactions in liquefied gases: e-addition tube; f-fritted glass; r-reaction flask.

An apparatus for reaction of solutions of alkali metals in  $NH_3$  with solid materials (G. Brauer and V. Stein [10]) is shown in Fig. 72. The alkali metal is sealed in a small ampoule which is provided with a small hook at one end. The seal point is broken off; the ampoule is suspended by the hook from a thin wire and introduced into the apparatus. The ampoule contents are melted in a stream of  $N_2$  and flow to B. Then  $NH_3$  is condensed in B. The solid is already in A, and both substances are combined by tipping the apparatus. After evaporation of the  $NH_3$  the reaction product is transferred to analysis flasks, Mark capillaries, or to containers such as shown in Fig. 54a. From these it can be further transferred, as required.

The Jander and Schmid [10] apparatus shown in Fig. 73 has also proven useful in such reactions. The first condensation of  $NH_3$  is greatly facilitated by the addition tube e; the frit f can be used for filtration in either direction. The greased ground joints remain gas tight even on immersion in a Dry Ice-methanol bath, so long as the pressure inside the apparatus does not fall below 50-100 mm. In a higher vacuum, fine channels form readily in the lubricant layer. An apparatus for liquid—liquid  $NH_3$  reactions yielding solid precipitates is described by Schwarz and Schenk and also by Schwarz and Jeanmaire [10].

A description of an apparatus for titrating liquid  $NH_3$  solutions with standardized  $NH_3$  solutions is given by Zintl, Goubeau and Dullenkopf [10]. Furthermore, similar devices for work with liquefied gases have been proposed by Juza, Schmitz-DuMont, F. Seel, G. W. Watt and others. A selection of the latest literature on the subject is given in [10]. The reader should also refer to the preparation of thiocyanogen in Part II, as well as to the section on Carbonyls and Nitrosyls in Part III.

# **Electrical Discharges**

One distinguishes here between the so-called "silent" discharges at atmospheric pressure and glow discharges at reduced pressure. Other discharge types such as arcs are not considered here, since in those cases the specific discharge effects are masked by thermal ones.

Silent discharges are obtained in a Siemens ozonizer, which is a system of two concentric tubes. The gas flows through the annulus (see Part II, section on Oxygen and Ozone).

Glow discharges are produced at reduced pressure (< 10 mm.) between electrodes (Al, Fe) connected to a high-power, high-voltage source (about 6000 v., 100-200 ma.). "Electrodeless" discharges can be used when metal electrodes cannot. They are produced by inserting the discharge vessel (a sphere or short cylinder) into a coil made of a few windings of thick copper wire (primary winding of a Tesla coil) and using the latter as an oscillatory circuit connected to a high-frequency generator (quenched spark gap, emitter). Especially effective are direct current pulse discharges, which are produced by charging a high-capacitance condenser from a highvoltage transformer via a rectifier and high-resistance rheostat. After reaching the break through voltage, the condenser discharges across the spark gap and the discharge tube with an extremely intense current surge. For a short time  $(10^{-5} \text{ sec.})$ , pulses of more than 100 amp. appear in the gap. Such apparatus is particularly suited for production of active nitrogen. A fuller description will be found in [11].

Aluminum foil is used for electrodes whenever possible. It is the least dust-forming material. In the case of halogens, water-cooled iron tubes (Schwab's apparatus [11]) must be used. The discharge tube is approximately 20 mm. I.D. and is enlarged at the end, where a 40-mm.-thick Al electrode is placed. If the electrodes are sufficiently close to the walls, the tube can carry a current of 200-300 ma. without special cooling. For higher currents a fan is used for cooling. Two such discharge tubes are shown in Fig. 74. Good pumps are especially important when working with electrical discharges, particularly when one of the reagents is a gas that does not condense at the temperature of liquid nitrogen. For example, for a throughput of one mole of gas per hour at 1 mm. pressure, 17,000 liters of gas, i.e., 5 liters per second, must flow through the apparatus. Since in many cases one must work with still lower pressures (0.3-0.5 mm.), very high capacity pumps as well as tubes of at least 20 mm. I.D. are absolutely necessary.



Fig. 74. Discharge tubes: D) sealing-off or cutting-off point (removal of reaction product frozen out in the trap); G) to the pump.

# **Purification of Substances**

Purification is effected in most cases by distillation, sublimation or recrystallization. In many cases, the mechanical methods of elutriation and gravity separation are also useful. The progress of the purification is followed by checking either the analysis or the physical properties, especially melting point, boiling point and vapor pressure.

#### DISTILLATION

Distillation columns are used for greater separation or to accelerate the process. The disadvantage of a more or less considerable holdup, which formerly required large quantities of substance in order to achieve effective fractionation, has been largely overcome by the development of modern columns. As a makeshift device, a 60-cm.-long column packed with metal or glass Raschig rings, and insulated by a glass jacket, can be used (Fig. 75). If the rate of distillation is adjusted so that for each drop collected at the top, two or three drops fall back into the boiler, then the separation is usually very good. Columns like Jantzen's or Podbielniak's are used when purity requirements are higher (Fig. 76). The principal component of the Jantzen column is a long, spiral



riser tube, which is thermally insulated by a sealed-off, silvered, evacuated jacket. The ratio of distillate to reflux is controlled by the setting of the stopcock (cf., e.g., Bernhauer, Einführung i.d. org. chem. Laboratoriumstechnik [Introduction to Organic Chemistry Laboratory Technique], Vienna, 1944, p. 119). The Podbielniak column is used for fractionation of liquefied gases. Columns with a rotating metal core (also available as microcolumns) have very small holdups.

#### VACUUM DISTILLATION

The usual equipment for vacuum distillation is so familiar that it requires no special mention here. Although the main use of short-path, thin-layer distillation is in preparative organic work, this method is also useful in preparative inorganic work. In this type of distillation, the substance runs in a thin layer over a heater

surface at extremely low pressure, and the more volatile components condense on a cooled wall directly above and a very short distance from the heater. A great many devices have been designed for this. One is presented in more detail in Part II, section on Sulfur, when dealing with the purification of polysulfanes.

#### SUBLIMATION

Besides the usual sublimation equipment (an example of the simplest type is a beaker covered with a water-filled flask) one should mention an apparatus for vacuum sublimation which can easily be assembled from a Pyrex evaporation vessel with a per-forated cover. Figure 77 shows the construction.

Vacuum sublimation is used, above all, for purification of metals. Except for a high-capacity pump (most metals give off

Fig. 75. Distillation column packed with Raschig rings.



Fig. 76. Podbelniak distillation column: a) filled with hydrogen at various pressures; b) high vacuum; for the sake of clarity, the diameters of all parts of the apparatus have been enlarged three times in relation to the length.

considerable quantities of gas upon heating) the equipment required is relatively simple. All that is required is a large quartz, ceramic or suitable metal (e.g., steel) tube into which the metal to be sub-

limed is placed, either directly or in a boat, and a concentric cold finger cooled with running water, on which the sublimed metal deposits. The cold finger must be easily removable at the end of the sublimation. Figure 78 shows the principle of the setup. It may be modified somewhat, depending on the materials used [e.g., quartz in the apparatus of W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. 219, 45 (1934)].

Less volatile metals can also be distilled, in smaller quantities, in a tungsten boat heated to avery high temperature. They deposit on an adjacent cooled surface. The layout corresponds



Fig. 77. Vacuum sublimation apparatus.



Fig. 78. Sublimation onto a cold finger.

very closely to the tubular tungsten furnace shown in Fig. 13. In this case, however, the heating element is an open boat, made by folding a sheet of tungsten, instead of the tube.

#### RECRYSTALLIZATION

Crystallization depends essentially on two factors; the crystallization rate and the number of centers of crystallization. With high supersaturation, the number of nucleation centers is large so that many small crystals are formed. With only slight supersaturation, the crystallization rate is controlling and only a fewand therefore larger—crystals are formed. Although the widely held notion that especially well-formed crystals are always highly pure is not valid (impurities can be occluded with the mother liquor), still, the production of large crystals for crystallographic purposes is so important that we shall discuss the subject briefly.

# CRYSTAL GROWING

Larger single crystals can be obtained by recrystallization from the gas phase, from a solution, from a melt and in metals [12].

Growing from the gas phase (sublimation) is effected by enclosing the substance in an oblong glass or quartz ampoule and maintaining a temperature gradient along the ampoule for some time. The transport phenomena that occur during this process lead to crystallized deposits. The process is widely applicable and is sometimes very effective, even in reversible decomposition reactions with participation of a gas phase. It is, however, suitable only for small quantities.

Growing from a solution occurs through growth on already present crystallization centers or on crystal seeds. The following general rules are valid. The crystallization vessel should either be very well insulated to achieve extremely slow cooling and a supersaturated solution at a given temperature, or the vessel should be maintained at a constant temperature (Dewar flask, thermostatted bath with slowly decreasing temperature) while the solvent slowly evaporates. Large crystals are obtained more easily in large diameter containers than in narrow ones. Crystallization centers on the bottom of the container disturb the growth of the crystal and should be avoided as much as possible, e.g., by polishing the surface. After each crystallization, the best crystals are selected for use as growth centers, and these are again inserted into a hot saturated solution. (However, the saturation should not be quite complete at the moment of insertion.) Some dissolution of the seed crystals in a new crystallization run is necessary for even growth. A single, well-shaped crystal can also be suspended in the solution on a thread or wire. Small crystals formed on the surface of the liquid should not be allowed to fall on the growing crystal; for this reason the latter is protected by a cover, made, for instance, of fibrous material.

Uniform growth is promoted to a large extent if either the crystal or solution is in continuous motion, since local concentration fluctuations are thus evened out. Freedom from vibration during cooling—previously considered important—is therefore not necessary in order to obtain large, well-formed crystals. Either the crystal is rotated in the still solution or the liquid is permitted to flow past the fixed crystal. For the first case, Johnson has devised a simple setup, which is shown in Fig. 79. A stream of air controls the evaporation rate of the solvent. There is a protective cover

over the growing crystal. For the second case, that of moving liquid, the simplest setup is an inclined, round-bottom flask which rotates slowly about its axis in a thermostatted bath with gradually decreasing temperature. With more stringent requirements, the Nacken apparatus (shown in Fig. 80) is used. Constant change of the solution, which is always being saturated in the middle section of this apparatus, is achieved by means of the two check valves and a rubber bulb attached at a.

When crystals are grown from a melt, the simultaneous growth of a great many crystallization centers must be prevented. The growing crystal should



Fig. 79. Single crystal growing with movement of the growing crystal.

always be the coldest part of the surroundings by being connected to a heat sink (a rod or a tube with good thermal conductivity).

An especially successful process for alkali halide crystals was devised by Kyropoulos [12]. The experimental arrangement can be seen in Fig. 81. The fused mass in the electrically heated crucible is heated to about  $150^{\circ}$  above the melting point, then cooled to about  $70^{\circ}$  above the melting point. The cooling rod is then immersed. Only then is the cooling of the rod started. When a crystal forms at the tip of the rod, usually as a hemisphere, the rod is carefully lifted by a micrometer device so that it barely touches the surface of the fused mass. At this point there develops a large, very clean, round crystal, provided the cooling of the rod is adequate. Finally this crystal is lifted from the mass and very carefully cooled.



Fig. 80. Single crystal growth in a moving solution.



Fig. 81. Single crystal growth in a melt.

The growing of metal single crystals can be carried out via several methods. Tammann and Bridgman [12] have devised an apparatus for slow solidification of metallic melts. A tube filled with the melt is lowered slowly and at a uniform rate (e.g., by means of a clock mechanism) through a vertical, electrically heated tubular furnace. In order to force the crystallization process to occur at a fixed place and from only one crystallization center, the bottom of the tube is drawn out to a capillary point (Schubnikow; Straumanis [12].

By pulling, a single crystal metal fiber is drawn out of the molten mass at a slow, uniform rate (e.g., 0.2 mm./sec.) while cooled and protected by a stream of inert gas (N<sub>2</sub>, CO<sub>2</sub>). A small mica leaf with a hole in the middle, floating on the surface of the molten mass, serves as a die for the fiber and determines the thickness of the wire-shaped single crystal (Czochralsky; Von Gomperz; Mark, Polany and Schmid [12]). Other special methods for growing metal single crystals are recrystallization with alternate mechanical deformation and annealing (systematically and successfully used till now with Al, Mg, and Fe) as well as a crystallization process in which a volatile metal compound decomposes thermally on a high-temperature filament and the metal deposits on it. This procedure may be used also for the preparation of some metals and very pure simple metal compounds. Some of these cannot be obtained as pure by any other method (Ti, Zr, Hf, Nb, Ta, etc.) (Koref; Van Arkel; Agte; Burgers; De Boer [12]).

Small crystals, required for powder pattern analyses, are often accidentally found in cavities of solidified melts. In some cases, this phenomenon can be artificially encouraged by inserting into the metal (or alloy) mass a honeycomb of folded strips of sheet iron, the nooks and crannies of which serve as cavities. In this case, the container must be moved to and fro during slow solidification, in contrast to other methods of crystal production (G. Brauer and R. Rudolph [12]).

Zone melting. In this method for metals and semimetals, developed by Pfann, a small heating device is passed along a rodshaped sample of a solid, fusible substance in such a way that a narrow fused zone gradually moves along the whole length of the sample. Impurities are thus transported in one direction. Several such passes result in ultra high purity material. The method is usable for all substances (and compounds) with appropriate melting properties [13].

# GRAVITY SEPARATION

Of the mechanical methods of separation, which can be used when chemical isolation of a product is either not possible or not practical, density separation requires some mention. A solid mixture of two components with different densities can be separated by means of a liquid of intermediate density. It is an essential prerequisite that the solid be pulverized until every particle is as homogeneous and free of inclusions as possible, but an unnecessarily fine powder should not be produced. Various "heavy liquids" and "heavy solutions" have been proposed as separation liquids, in which the lighter component rises and the heavier one sinks to the bottom. If somewhat elevated temperatures are permitted, then low-melting substances can be included in this group and the range of the usable density values widened a little more. Table 16 shows the substances used most often for gravity separations.

The following gravity separation procedures are of importance [14].

Dense liquids and solutions. In the simplest case, narrow, tall beakers or ordinary graduated cylinders are used; these are filled

with liquid and the solid is stirred in with a glass rod and left to separate. The light component can be decanted after settling or, better still, scooped out with a nickel wire sieve. Ordinary separatory funnels whose stopcock is mounted a few centimeters below the vessel are also recommended. The bore of the stopcock plug should be the same diameter as the funnel tube. For more thorough separations an apparatus such as shown in Fig. 82 I can be used. The mixture to be separated is poured into the large tube and liquid is allowed to run through the funnel until its level equals that of the upper side branch. By repeated stirring with a glass rod, the light components are elutriated toward the top. Then the main large opening is closed off with a cork and more liquid poured in through the funnel. This liquid sinks to the bottom, raises the lighter components to the top, and thus permits them to overflow. The operation should be repeated several times. For other recommended apparatus, see [17].



Fig. 82. Apparatus for gravity separation. I) after Brauer and Scheele; II) after Penfield.

Dense melts. The separation can be effected in a test tube, which after solidification can be broken and the cake split up into two parts containing the lighter and the heavier components. For more separation, Penfield's apparatus, shown in Fig. 82 II, is used. A small container b is placed in a Vycor or quartz test tube a. Vessel b is connected by a ground glass joint to an adapter c. The resulting tube (b plus c) is filled with a mixture of the powder to be separated and the substance serving as the melt. Then the entire vessel (a, b and c) is heated, e.g., in a water bath; the molten substances will rise fully into c. After separation has been accomplished, the closure tube d, provided with a male ground joint at the bottom, is introduced to seal off b, as shown in Fig. 82 (II). The spatial separation of dense and light substances is so good that after solidification of the melt, b can be loosened without difficulty from c and d by brief heating.

# Table 16a

# Liquids for Gravity Separation. Dense Liquids [15]

Liquid	d <sub>max</sub>	Properties
Bromoform	2.9	Cheap, very mobile, transparent, inert to ores, slightly sensitive to light
lodoform solution in bromoform	2.9-4.0	Similar to bromoform, but strongly colored to the point of capacity. At high CHl <sub>3</sub> con- tents (m.p. 119°C) solid at
Tetrabromoethylene	3.0	Similar to bromoform; more light- stable than bromoform. Evapo-
Potassium iodomercurate solution (Thoulet solution)	3.2	Almost colorless, transparent, easily prepared. Poisonous! Attacks the skin! Hygroscopic, fairly viscous; decomposed by Fe and many oxides and sulfides, with Hg separation. Crystal powder removal difficult be- cause of adsorption
Methylene iodide	3. 32	Pure material almost colorless; very mobile; easily washed with benzene; fairly inert when pure; sensitive to sunshine, heat and less noble metals (Al). Rela- tively experience.
Cadmium borotungstate solution (saturated) (Klein solution)	3, 36	Not poisonous, harmless; difficult to prepare; fairly viscous; de- composed by Fe, Zn, Pb and carbonates
Barium iodomercurate solution (saturated) (Rohrbach solution)	3.59	Easily prepared; cannot be de- composed by carbonates; should be diluted with KI solution
Thallium formate solution (saturated)	3 <b>.</b> 17 (12*C)	Densest known aqueous solution; mobile; may be diluted with water. Light vellow when
	4.76 (90°C)	freshly prepared, becoming brown upon standing; the brown color can be removed with
Thallium formate and malonate solution (1:1) (saturated) (Clerici solution)	4.07 (12°C)	charcoal by heating the diluted solution. Somewhat difficult to prepare; marked changes in
	4.65 (50°C) 5 (95°C)	density with temperature and evaporation.

# Table 16b

# Dense Melts [16]

Melt	M.p.	d	Properties		
Silver nitrate	198°C	4.1	Wide variety of uses. Can be diluted with KNO2 or NaNO2.		
Mercury(I) nitrate	70°C	4.3	Decomposes on long melting; thus, addition of several drops of concentrated HNO <sub>3</sub> is advantageous. Risk of decomposition with free metals.		
Thallium nitrate Thallium nitrate- silver nitrate (in ratios depending on the powder to be separated) (Retgers)	206°C 75°C	5.3 4.5 to 4.9	Prepared by dissolving proper amounts of Tl and Ag in HNO <sub>3</sub> . Can be diluted with water, with strong melting point depres- sion. Colorless, mobile, transparent liq- uid. An Ag film forms under the influence of strong light. This can be redissolved with a little HNO <sub>3</sub> . Sulfides decompose the melt. Silicates are partially attacked.		
Thallium mercury(I) nitrate (Retgers)	76°C	5.3	Best heavy melt; highly mobile; clear solu- tions with $H_2$ O in all proportions. Also suitable for sulfides and silicates.		

In all separations with dense melts, care should be taken to maintain a constant temperature.

# Analysis of Purity

Although melting point determination is of special importance for organic analysis, it is often useful in inorganic preparative work as well. Of course, the usual methods, using a capillary in a thermostatted bath, cannot be used in most cases because of the high melting points. Higher temperatures can be reached with a copper or aluminum block [18]. For high temperatures, the crucible method is used, in which the cooling rate of the melt is followed with a thermocouple immersed in it. The method is greatly simplified by adding to the setup a second thermocouple placed next to the first and connected in a circuit with a third one placed in a second crucible, which stands beside the melt and contains a comparison substance (e.g., fine sand) (see Fig. 83, differential thermocouple). As long as the melt and comparison crucibles are at the same temperature, the two thermoelectric currents cancel. As soon as a temperature difference develops as a result of the heat of crystallization, the highly sensitive instrument deflects and the temperature corresponding to the first crystallization is read off from the proper thermocouple. This method of differential thermal analysis is, in addition to melting point determination, also quite generally suitable for the investigation of all changes, reactions, etc., which are accompanied by thermal effects.



Fig. 83. Determination of solidification point with differential arrangement of the thermocouples

For very high temperatures, or if only verysmall amounts of material areavailable, Burgess's micropyrometer can be used [18].

Measurement of vapor pressure of lowboiling substances has already been treated in the discussion of Stock's apparatus. With substances boiling above room temperature. a heated manometer must be used. For instance, molten tin can be used as the manometer liquid and the flask containing the substance together with the short manometer lowered into the heated bath, while the second leg of the manometer is connected to a Hg manometer and a surge vessel. The pressure is then compensated for until the tin in both legs is at the same level, and the pressure is read off on the Hg manometer. Another very useful device is the isoteniscope of Smith and Menzies [18]. Here the substance itself serves as the manometer liquid. Figure 84 shows the apparatus. The





substance is poured into the small flask  $\alpha$  and the instrument is evacuated; the material is permitted to boil a little and the instrument is then tilted so that a portion of the substance flows

into the manometer leg. Then the entire instrument is brought to the desired temperature and the pressure is so controlled that the liquid levels in both legs are the same. The pressure is then read off on the mercury manometer.

The best device, however, is the quartz coil manometer, the coil of which can be heated to  $500^{\circ}C$  (in special cases to  $600-700^{\circ}C$ ). In all cases the null point of the instrument must be checked after each measurement. Therefore the manometer should be provided with a heating coil, which does not need to be at the test temperature but must nevertheless be at a sufficiently high temperature to prevent condensation in the coil and in the capillary connections (which are likewise provided with a heating coil). With compensation to zero, the pressure is read off on the Hg manometer. In those cases where it cannot be ascertained by the usual method (with a thermometer and distillation flask) the boiling point is determined more accurately by extrapolation of the vapor pressure curve.



Fig. 85. Vapor pressure eudiometer.

Vapor pressure eudiometer. This apparatus was originally devised as a tool for analytical checking of the course of decomposition of ammines or hydrates, but it is also useful for preparatory purposes, e.g., when determination of a definite stage of decomposition is desired. Figure 85 shows the construction of this device. The substance is enclosed in the smallest possible flask a and a definite quantity of the volatile component is removed from it. The right leg of the manometer m is calibrated in milliliters from the zero position (both legs at equilibrium) down. The volumes of a, of the various tube sections between stopcocks  $h_1$ ,  $h_2$  and  $h_3$ , and of the auxiliary flasks b, c and d are measured partly by direct weighing and partly determined indirectly, using the gas laws. The volatile component from a is removed intermittently and measured by means of auxiliary flasks b and c. The weight decrease of the substance is followed by weighing flask a, which is closed off by  $h_1$ ; in order to be able to remove a at  $h_1$ , dry air can be introduced at  $h_2$  (G. F. Hüttig; G. Jander and H. Mesech [18]).

# **Powder Reactions**

A certain mobility of the participating reagents as well as the largest possible contact surface are necessary for reactions between two solid substances. The mobility (rate of diffusion) and reactivity can be enhanced by raising the temperature or raising the energy content of the material by fine changes in the structure. According to a rule first advanced by Tamman—which is only approximately valid—reasonable conversions are obtained in normal experimentation times (order of magnitude: 1-100% conversion, 0.1-100 hours) only when at least one of the solid reagents is heated to over 2/3 of its absolute melting temperature [example: conversion of  $Al_2O_3$ , m.p.  $2320^{\circ}$ K, is expected to be rapid only above  $1550^{\circ}$ K (about  $1300^{\circ}$ C)].



Fig. 86. Press forms for powder compression.

A large common surface of the reagents, which favors diffusion, is ensured by using a fine powder. The powders should be completely mixed and mechanically compressed into briquettes. A simple apparatus for the production of such briquettes (tablets) is shown in Fig. 86a. Such devices may be easily built in the laboratory from iron rods and are also available commercially (e.g., to make samples of combustible substances for calibration of calorimeters or KBr tablets for IR analysis). In simple cases, a screw press of suitable size suffices; for large cross sections and for high pressure, hydraulic presses are used. The contamination of the surface of the pressed object by traces of iron from the mold wall due to abrasion can scarcely be helped. If this is very troublesome, the surface of the finished pressed object can be ground or scraped off. Alternatively, a glass tube can be tightly fitted into the brass mold and the powder compressed with glass pistons inside this glass tube. Such an apparatus consisting of a sheathed glass tube (I.D. 5-6 mm., O. D. 7-8 mm.) can be operated at pressures of 100 kg./cm.<sup>2</sup> without breaking.

Figure 86b shows a mold that can be disassembled, devised by G. Grube and H. Schlecht [Z. Elektrochem. 44, 367 (1938)]. It is made of machined steel pieces and produces oblong, columnar wedges of compressed powder.

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# Part II Elements and Compounds

# SECTION 1

# Hydrogen, Deuterium, Water M. BAUDLER

# Hydrogen

Нz

Commercial hydrogen, available in steel cylinders, is produced either by electrolysis or by the water shift reaction from water gas.

Electrolytic hydrogen contains 99.7-99.8%  $H_2$ . The only impurity is air, with the oxygen amounting to less than 0.1%. This commercial hydrogen may be treated either by passage through a combustion tube filled with reduced CuO wire at 400°C, or by passage through the "active copper tower" of Meyer and Ronge (see section on Nitrogen), followed by drying with CaCl<sub>2</sub> or P<sub>2</sub>O<sub>5</sub>. The gas obtained by either of these methods may be used for most laboratory applications, since its very small N<sub>2</sub> content (about 0.2%) is usually not harmful. If commercial electrolytic hydrogen is unavailable, it may be prepared in the apparatus described in the section on Nitrogen (the polarity is reversed, compared to oxygen preparation!).

On the other hand, commercial hydrogen produced from water gas is contaminated with considerable amounts of CO, CO<sub>2</sub>, O<sub>2</sub> and N<sub>2</sub>, and sometimes also with AsH<sub>3</sub> and Fe(CO)<sub>5</sub>. The CO<sub>2</sub> may be removed by absorption with KOH or soda lime; the AsH<sub>3</sub> is taken up by a fully saturated KMnO<sub>4</sub> solution (containing solid KMnO<sub>4</sub>). The O<sub>2</sub> is separated out either by passage over heated copper wire or over red-hot Pt-asbestos (prepared according to the directions given in the section on Platinum Metals). The latter procedure also results in thermal decomposition of the Fe(CO)<sub>5</sub>. The removal of CO is more difficult, since neither absorption in acid or ammoniacal CuCl solution nor oxidation with HgO (or HIO<sub>3</sub>) is quantitative. The most reliable method of removing CO is freezing out at the temperature of liquid nitrogen. In any case, pure H<sub>2</sub> is best prepared from electrolytic hydrogen.

Very pure, completely air-free hydrogen may be prepared by any of the following methods.

# I. HEATING OF PALLADIUM SPONGE

Palladium sponge, prepared by the reduction of  $PdCl_2$  solution (see section on Platinum Metals), is carefully washed with hot water, dried and well calcined by heating with a burner. The hot product is charged into a preheated combustion tube (provided with a manometer sealed to one end) and allowed to cool slowly in vacuum. When the sponge reaches room temperature, a carefully prepurified and predried  $H_2$  stream is admitted into the tube and is to a large extent absorbed by the Pd. The absorption produces a slight glowing of the sponge. When the sponge is then heated to about 200°C, pure  $H_2$  is liberated. A steady stream of the gas may be obtained with the aid of a small pump. In this way, 100 ml. (STP) of  $H_2$  may be obtained per gram of palladium.

This method is especially useful in the preparation of small quantities of very pure hydrogen. E. von Angerer (Technische Kunstgriffe bei physikalischen Untersuchungen [Industrial Techniques Applied to Physical Experiments] 6th ed., Braunschweig, p. 92) shows an apparatus capable of continuous production of 100 ml. of pure  $H_2$  per hour. It operates on the principle of hydrogen diffusion through electrically heated Pd tubes.

# II. DIFFUSION THROUGH NICKEL

Commercial hydrogen may also be further purified by diffusion through nickel. This may be accomplished in the apparatus shown in Fig. 87, which yields a steady stream of very pure gas at atmospheric pressure.

The basic component is a pure nickel, precision ground, seamless tube (diameter 2 mm., length 5 m., wall thickness 0.1 mm.) soldered shut at one end. Five such tubes are needed. Each is coiled into a helix, the helices are intertwined, and the open end of each tube is soldered to a brass header, as shown. The header is provided with a standard tapered male joint n. For ease of handling, the tubes are heated in a H<sub>2</sub> stream at 1000°C for two hours, after which they may easily be bent by hand. The helix assembly is inserted into a quartz reactor tube 1 m. long and 35 mm. in diameter. The front and back headers are cemented to the quartz tube with picein, as shown. The back header is provided with a needle valve v, which serves for fine control of pressure in the tube and through which the gaseous impurities, which are contained in the hydrogen and which accumulate in the reactor, may be released and subsequently burned. The front header has a connection for a mercury manometer. Only the middle part of the quartz reactor is electrically heated. Thus, the soldered points of the nickel tubes remain in the cooler sections of the apparatus.



Fig. 87. Purification of hydrogen by diffusion through nickel: v is a needle valve for fine control of pressure in the apparatus.

Depending on the operating conditions, the apparatus is capable of delivering the following quantities of hydrogen:

°C	15	20	25	30	mm. Hg
750	20	27	34	41	ml./min.
815	27	36	43	52	
860	34	45	55	68	
900	41	54	68	84	

The gas output is proportional to the pressure in the reactor but does not vary linearly with temperature. By varying the pressure, any desired gas output can be obtained almost instantaneously. Provided the feed gas cylinder has a good regulator, the reactor will give trouble-free operation for about 250 hours. However, it must be tested for leaks before each run.

III. DECOMPOSITION OF UH<sub>3</sub>

$$\begin{array}{l} 2 \text{ UH}_3 = 2 \text{ U} + 3 \text{ H}_2 \\ 482.19 & 476.14 & 6.05 \end{array}$$

This procedure permits the production of very pure hydrogen free of noble gases. The gas may be obtained in any desired amount and at any time from previously made  $UH_3$ .

The UH<sub>3</sub> may be prepared in the apparatus shown in Fig. 88. Commercial electrolytic hydrogen (from a cylinder) is prepurified by passage over copper shavings in tube b at 650-700°C and drying with anhydrous Mg(ClO<sub>4</sub>)<sub>2</sub> in tube c. The gas may be further purified at d by passage through pulverized uranium at 700-750°C. This prepurified hydrogen may then be converted to  $UH_3$  in the two-neck flask f, which is half filled with uranium turnings. These turnings must also be prepurified by treatment with dilute  $HNO_3$  (to remove the oxide film), washing and drying. Flask f is heated either with a nitrate-nitrite salt bath or an electric furnace. The temperature in the flask is 250°C. Two wash bottles, one empty and one filled with concentrated  $H_2SO_4$ , are attached to flask f.



Fig. 88. Preparation of uranium hydride and purification of hydrogen. b) tube filled with copper shavings; c) tube filled with  $Mg(ClO_4)_2$ ; d) tube containing uranium powder supported and covered by glass wool plugs e and e'; f' flask with uranium turnings; g) heating bath;  $s_1, s_2$  ground glass joints.

The apparatus must be thoroughly purged with hydrogen prior to the run, i.e., prior to heating b, d and f. The reaction is completed when the  $H_2SO_4$  in the riser of the last wash bottle is no longer pulled upward by suction upon interruption of the  $H_2$ stream.

The UH<sub>3</sub> product is a brown-black, spontaneously igniting powder. Very pure hydrogen may be liberated from it by heating, possibly at reduced pressure, to 400°C (or to a somewhat lower temperature). The uranium powder residue remaining after the decomposition reacts vigorously with H<sub>2</sub> at room temperature. The reaction is still quite vigorous at  $-80^{\circ}$ C and ceases only at  $-200^{\circ}$ C.

# IV. DECOMPOSITION OF TITANIUM HYDRIDE

Titanium hydride is well suited for the production of larger quantities of very pure hydrogen. It has a relatively low decomposition temperature ( $400-900^{\circ}C$ ), a relatively high hydrogen content, and is easily regenerated. Aside from this, titanium oxide and nitride are completely stable at the required decomposition temperatures. The decomposition is endothermic. Thus, the evolution of gas ceases whenever the flow of heat is reduced, and a continuous, well-controlled gas stream is obtained. It is advisable to use the apparatus shown in Fig. 89, so that the very pure  $H_2$ product may be immediately used in hydrogenation reactions, which may be conducted in the space provided at g.



Fig. 89. Preparation of very pure hydrogen from titanium hydride. a) quartz reactor tube; b) molybdenum boat containing Ti; c) heating winding; d) radiation shield; e) radiation shields for protection of stoppers;  $\mathcal{N}$  glass wool;  $\mathcal{Q}$ ) apparatus for conducting reactions with the very pure hydrogen product; the tube contains a boat for the reactants and is surrounded by an electric furnace. This part may be omitted if the hydrogen product is to be used elsewhere;  $\mathcal{M}$  pressure-sensing switch; r) relay.

A quartz reactor tube a (O.D. 34 mm., I.D. 30 mm., over-all length 1500 mm.) is wound over a length of 650 mm. with a heating coil c, which is cemented to the tube with a thin quartz-waterglass slurry. Molybdenum boat b is placed in the heated zone. The radiation shield d retards heat loss to the outside. Switch m controls the heat input to the winding, sensing the pressure developed by the hydrogen product stream.

The titanium hydride is prepared as follows: commercial titanium sponge of usual purity and medium grain size is placed in the molybdenum boat and dried in a stream of commercial hydrogen at 400°C. This step may sometimes be omitted. Following this, the temperature is raised to  $700^{\circ}$ C. The material is then heated for 30 minutes while maintaining the gas flow. Then, after thorough evacuation of the apparatus, the product (titanium hydride) is heated

until a pressure of 0.1 atm. gauge is registered on switch m, at which point the current is shut off. The pressure decreases due to the rapid drop in temperature and consequent gas volume contraction (and/or use of the gas for hydrogenation at g). When the control point pressure is reached, the current is again switched on. Despite this simple "on-off" control, pressure fluctuations are small.

After the desired amount of hydrogen has been liberated, the titanium hydride may be regenerated by heating in commercial hydrogen and subsequent cooling. A charge of 500 g. of titanium sponge will liberate 100 liters of pure  $H_2$  per run.

# V. ELECTROLYSIS IN THE ABSENCE OF AIR

An apparatus for electrolytic preparation of  $H_2 \text{ or } O_2$  (depending on polarity) in complete absence of air is described in the section on Nitrogen. The product gas contains less than  $4 \cdot 10^{-6}\%$  air.

#### **PROPERTIES:**

Formula weight 2.016. Colorless, odorless, tasteless gas. Its reducing action is especially apparent at high temperatures. For this reason, hot  $H_2$  should not be passed through concentrated  $H_2SO_4$ , since it then becomes easily contaminated by  $SO_2$ . M.p. -259.2°C, b.p. -252.8°C;  $t_{\rm CT}$  -239.9°C,  $p_{\rm CT}$  12.8 atm. gauge; d (liquid) 0.070; weight of 1 liter  $H_2$  at STP = 0.08987 g. Solubility in water at 760 mm.:0.021 vol./vol. at 0°C, 0.018 vol./vol. at 20°C, 0.016 vol./vol. at 100°C. Solubility in other liquids is also very small.

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# **Pure Water**

The usual laboratory distilled water contains considerable amounts of dissolved  $CO_2$  and, occasionally, traces of  $NH_3$  and organic substances.

This distilled water may be purified with  $CO_2$ - and  $NH_3$ -free air, which is allowed to bubble through at 90°C for 24 hours. The air should be drawn from outside the building, since laboratory air is often quite badly contaminated. Before contacting the water, the air passes successively through a wash bottle filled with concentrated  $H_2SO_4$ , two bottles with NaOH, and one filled with pure water. Avoid long rubber tubing connections.

This prepurified water is then doubly distilled, first with addition of some NaOH and  $KMnO_4$  and then in the presence of a small quantity of KHSO<sub>4</sub>. The condenser and its connections should be of Sn, Pt or quartz. Glass condensers must be avoided. It is advisable to bend the condenser outlet at a right angle and insert the leg directly into the neck of the receiver, using no sealing materials (see Fig. 90). To avoid condensate spraying, a vapor trap is installed before the receiver, as shown in the figure (b).



Fig. 90. Distillation of pure water. Adapters for transition from condenser to the receiver; a) simple and inexpensive; b) with a vapor trap.



Fig. 91. Distillation of "conductivity" water. ST are standard ground joints.

The receivers should be of quartz, Pt or Pyrex and must be thoroughly steamed out before use. The distillation should be slow and large volumes of forerun and residue should be discarded. Contact with laboratory air should be avoided as much as possible.

The product may be tested for purity by the conductivity method. The freshly distilled product should have a conductivity of about  $10^{-6}/\text{ohm}^{-1}\text{cm}^{-1}$ . It may be tested for CO<sub>2</sub> with Ba(OH)<sub>2</sub> solution and for NH<sub>3</sub> with Nessler's reagent.

Very pure water is stored in quartz or platinum containers. Pyrex vessels may be used, if properly steamed out and if employed only for water storage. The receiver neck should have a male ground joint and be closed by a cap with a female joint.

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"CONDUCTIVITY" WATER

Extremely pure water for conductivity measurements is obtained through very careful distillation of already thoroughly purified material. This prepurified water (conductivity at  $25^{\circ}$ C:  $1-2 \cdot 10^{-6}$  ohm<sup>-1</sup>) is obtained either via the method described above or through another double distillation procedure [the first distillation with KMnO<sub>4</sub> + H<sub>2</sub>SO<sub>4</sub>, the second with Ba(OH)<sub>2</sub>, using a Pyrex apparatus with a tin condenser].

I. Single-step distillation according to the method of Kortüm is done in the apparatus shown in Fig. 91. Except for the short quartz condenser, the apparatus is made of Pyrex. All connections are made with ground joints, except where indicated. The section between the reflux condenser and the quartz condenser is wound with a 60-ohm heating coil and heated to  $100^{\circ}$ C in order to avoid creepover of liquid water. The Pyrex reflux condenser is of the internal helix type. A ground joint adapter connects the condenser and the receiver. This adapter and the receivers must be thoroughly presoaked in hot, dilute acids (several days) to remove any impurities which may increase the conductivity of the product.

The pure water charge is distilled in a stream of air. Compressed air from a cylinder flows at a slow rate of 1 bubble/second through seven wash bottles. In succession, these are filled with concentrated  $H_2SO_4$  (1 bottle), 50% KOH (3 bottles) and "conductivity" water (3 bottles, preferably with glass frits). The same compressed pure air is used to transfer the product water from the receivers to storage vessels. The three grids heating the distillation flask consume about 300 watts. In order to improve the rate and uniformity of heat transfer, the space between the heating grid and the distillation flask is filled with ceramic beads. The center tube of the distilling flask permits charging and emptying the contents.

A conductivity cell is attached to the three-way stopcock at the outlet of the condenser. The distillate is discarded until its conductivity matches the desired value. Only then is the system connected to the receiver.

The apparatus delivers 100 ml./hr. of water having a  $\times$  (25°C) =  $2 \cdot 10^{-7}$  ohm<sup>-1</sup>. At very low distillation rates, water with a  $\times$  (25°C) =  $10^{-8}$  ohm<sup>-1</sup> may be obtained.

II. "Conductivity" water with  $\varkappa$  (25°C) =6-8 · 10<sup>-8</sup> ohm<sup>-1</sup>, in volumes larger than those provided by the apparatus of method I, can be obtained with the installation of Thiessen and Herrmann. This two- or three-step distillation does not require excessively complex equipment and is capable of delivering 400 ml./hr. of product.

**REFERENCES:** 

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"pH-PURE" WATER

The method for obtaining large quantities of water with pH = 7.00 is based on addition of NaOH and  $KMnO_4$  during the first distillation and  $H_3PO_4$  (to combine the NH<sub>3</sub>) in the second distillation step. A third distillation in quartz apparatus (to remove traces of alkali) follows.

REFERENCE:

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# **Deuterium and Deuterium Compounds**

Deuterium and the simpler inorganic deuterium compounds are commercially available. Nevertheless, the research chemist may occasionally be called upon to prepare some of these compounds, starting with  $D_2O$ —the most available of the deuterium compounds.

Heavy water is manufactured in concentrations ranging from 5 to 99.5%  $D_2O$  and is sold in sealed glass ampoules. Pure heavy water is very hygroscopic; i.e., it loses  $D_2O$  vapor while simultaneously absorbing air moisture. Therefore, certain precautions must be taken when filling or emptying  $D_2O$  ampoules.

If only a portion of the ampoule content is to be used, the pointed end of the ampoule is heated in a small flame and drawn out to a capillary with tongs. The capillary end is then broken off and the desired quantity of  $D_2O$  driven out by gentle heating,
e.g., by hand. The receiver is closed off as soon as possible and the ampoule is immediately resealed with a flame. It is best to store it in a desiccator.

The  $D_2O$  contents of an ampoule may be preserved from contact with air moisture and still utilized only partially in the following way: the entire contents of the ampoule are transferred by the method given below to an elongated flask, closed off by a piercable, membrane-type rubber stopper, such as used for serum vials. Then the desired amounts of  $D_2O$  can be withdrawn from the closed flask by means of a hypodermic syringe and injected into other vessels, which can also be closed off with the same type of stopper. The very fine capillary produced in the rubber stopper by the needle closes immediately upon withdrawal of the latter.

If the entire contents of an ampoule are to be used in a reaction, it is best to break and empty it inside the reactor itself, thus avoiding transfer operations. To accomplish this, the ampoule is placed in a snugly fitting vessel, such as shown in Fig. 92. This vessel is then melt-sealed to the reactor. The apparatus is then connected to a high-vacuum system. If avoidance of dilution of the  $D_2O$  content is critical, the entire apparatus is heated by fanning with an open flame to remove the film of "light" water accumulated on the internal surfaces. The vacuum is then disconnected, the apparatus is closed off, and the vessel containing the ampoule is rapidly immersed in liquid nitrogen. The sudden freezing of the  $D_2O$  causes the ampoule to shatter. Cooling with an acetone-Dry Ice mixture is not sufficient, because the solidification of the  $D_2O$  tends to be slower and its crystals begin to grow mostly in the upper, empty part of the ampoule. Alternatively,



Fig. 92. Breaking D<sub>2</sub>O ampoules by freezing with liquid nitrogen. the ampoule may be broken by a sudden movement of a glass-enclosed iron bar, suspended inside the reactor and set in motion by an electromagnet.

All substances to be reacted with  $D_2O$  must be carefully freed of all traces of water. Hygroscopic compounds, in which the uptake of small amounts of  $H_2O$  during charging of the reactor is unavoidable, must be re-dehydrated in the reactor itself. This is done by heating (in high vacuum, if possible), distillation or resublimation, where applicable. Again, such hygroscopic compounds may be enclosed in sealed glass ampoules immediately after their preparation. These ampoules may then be inserted into the reactor and broken with a magnet-operated iron bar, as described above. As far as possible, the apparatus should have fused connections and contain a minimum of stopcocks. If this is not possible, special care should be taken in sealing all possible leaks. Drying tubes should be inserted between the apparatus and its connections to the pumps (vacuum) or to the atmosphere. Better still, liquid-nitrogen-cooled gas traps should be used to prevent entrance of atmospheric moisture. Since in the presence of  $H_2O$  most inorganic D compounds exchange part of their D content for H, these precautions must be observed in all reactions described in later sections.

Large amounts of deuterium compounds are expensive. It is therefore advisable to practice each reaction with "light" starting material before attempting to use the D compounds.

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### Deuterium

## $\mathbf{D}_2$

I.

 $2 D_2 O + 2 Na = D_2 + 2 NaOD$ 40.06 45.99 4.03 82.02

Flask E of the glass apparatus shown in Fig. 93 contains an Al crucible with excess metallic sodium. Vessel V contains the  $D_2O$  reagent. The latter is introduced (as described above) in the absence of atmospheric moisture. After cooling V with liquid nitrogen, the apparatus is carefully evacuated, with stop-cocks 1 and 2 open. Stopcock 2 is then closed and the  $D_2O$  is distilled slowly onto the Na by cooling E with liquid nitrogen. To complete the reaction, E is then heated for several hours to 350°C. After opening stopcock 2, the  $D_2$  produced is transferred for purification into a receptacle filled with degassed charcoal and left there for some time at -196°C. If fresh Na is used, the  $D_2$  product will still contain some few percent of  $H_2$  after the purification (this  $H_2$  was dissolved in the metal and existed as NaOH). Pure  $D_2$ , containing less than 0.2%  $H_2$  and other foreign gases, is only obtained in the second run with the same piece of Na.



The gas is tested for purity by measurement of the thermal conductivity or vapor pressure. The yield of  $D_p$  is quantitative.

The method is especially suitable for the preparation of small amounts of  $D_2$  (up to one liter).

Π.

 $D_2O + Mg = D_2 + MgO$ 20.03 24.32 4.03

In an elongated flask of a Pyrex apparatus, pre-evacuated to  $10^{-4}$  mm., 20 g. of  $D_2O$  is slowly evaporated. The vapor passes through the reaction tube, set vertically on top of the flask. The tube (I.D. 2.4 cm. and 55 cm. long) contains 130 g. of Mg shavings of various sizes, with coarse particles on the bottom and loose powder on top. The column filling is supported by a perforated platinum disk which rests on glass lugs inside the tube. The Mg is heated to 480°C by a tubular furnace.

After an extended period of time, some magnesium silicide will form on the walls of the heated glass tube. To avoid this, it is suggested that the Mg be placed in a tube of unglazed hard porcelain which is then inserted into a Pyrex or Vycor tube and melt-sealed to the latter at one end. With such an arrangement the Mg may even be heated to a somewhat higher temperature and its reactivity thus enhanced.

For purification, the  $D_2$  product passes through a trap filled with glass wool and kept at -196°C. It is taken out from the generator as quickly as possible, either by condensation with liquid  $H_2$  or by forcing it into an attached storage container. An in-line flowmeter and a manometer allow constant checking of pressure. The rate of evolution can be controlled by varying the supply of heat to the  $D_2O$  flask. A maximum flow of 0.5 mole of  $D_2$ /hour may be obtained. Since the first  $D_2$  fraction may be contaminated with some  $H_2$  from the Mg and from the apparatus walls, it is advisable to collect some of the first  $D_2$  fraction in a separate vessel. The  $D_2$  formed later is very pure. The yield is quantitative.

This method allows rapid production of large amounts of  $D_2$  and utilizes the entire D content of the heavy water.

a)  $2 D_2 O + U = UO_2 + 2 D_2$ 40.06 238.07 270.07 8.06

b) 
$$3 D_2 + 2 U = 2 U D_3$$
  
12.09 476.14 488.23

This method is especially useful in that it makes possible both the preparation (Eq. a) and the storage (as  $UD_3$ —Eq. b) of  $D_2$ . High-purity  $D_2$  can then be liberated by thermal decomposition of the  $UD_3$ . Any desired quantity of very pure  $D_2$  can thus be obtained when needed.

The highly endothermic reaction of  $D_2O$  vapor with U may be carried out slowly and safely in the apparatus of Fig. 94. The 50-ml. flask a is connected by stopcock  $h_1$  with manometer b and quartz reaction tube d. Reactor d is heated with an electric furnace to 600-700°C and is connected to a liquid-nitrogen-cooled trap f. The latter is, in turn, connected to a high-vacuum pump and a flask g which may be heated to 250°C.

Flask a is about half filled with D<sub>2</sub>O and d and q with uranium shavings (the uranium is pretreated with dilute HNO<sub>3</sub> to remove all oxide, then washed and dried). The shavings in reactor d are supported on and covered with glass wool plugs  $c_1$  and  $c_2$ . The  $D_2O$  in  $\alpha$  is then frozen with a Dry Ice-methanol bath; this must be done slowly to avoid cracking the flask. The entire apparatus is then evacuated, while g and d are heated. The D<sub>2</sub>O is then carefully melted and the reaction is slowly started by allowing the vapor to penetrate to the uranium in d. The first  $D_2$  evolved is used to flush the apparatus, with stopcock  $h_3$  closed. Only then is  $h_{\lambda}$  closed and  $h_{\lambda}$  opened. During the reaction, a is kept at about 30°C. The D<sub>2</sub> product passes through trap f, in which any entrained traces of  $D_2O$  are frozen out, and is absorbed by the uranium shavings in  $\mathcal{G}$ , forming UD<sub>3</sub>. When all of the U is finally converted to  $UD_3$ , the excess  $D_2$  causes an increase in pressure which suppresses the evaporation of D<sub>2</sub>O and thereby prevents any further D<sub>2</sub> formation. Thus, once started, the process is selfregulating and requires no special attention. Several grams of  $D_2O$  can be converted into  $UD_3$  in one hour.

The  $UD_3$  is a brown-black, spontaneously igniting powder. To prepare very pure  $D_2$ , it is thermally decomposed either at atmospheric or reduced pressure (see also  $H_2$  above: III). The U



Fig. 94. Preparation and storage of deuterium. a)  $D_2O$ reservoir; b) manometer;  $c_1, c_2$ ) ground joints; d) quartz tube containing U turnings; f) trap; g) reaction flask containing U turnings;  $h_1-h_4$ ) stopcocks.

powder thus formed (at  $400^{\circ}$ C or lower temperatures) reacts vigorously with H<sub>2</sub> (or D<sub>2</sub>) at room temperature and still quite vigorously at  $-80^{\circ}$ C. Only at  $-200^{\circ}$ C does the reaction cease.

### IV. ELECTROLYSIS OF D<sub>2</sub>O

An electrolytic cell, holding 60 ml. of liquid and made from a standard ground glass joint, is shown in Fig. 95. The male part of the ground joint continues into a cylindrical water jacket (only partly shown in the diagram) which surrounds the cathode. The Pt electrodes are also cylindrical and are prepared by fusing together a Pt wire with a Pt foil. The  $D_2O$  electrolyte is acidified with 25%  $D_2SO_4$ . (If no  $D_2SO_4$  is available, carefully dehydrated  $K_2SO_4$  or  $Na_2CO_3$  can also be used.) After evacuation of the cell at A and B, electrolysis is begun at a low current to prevent foaming at low pressures. After a short time, however, the current can be increased to 5 amp. The temperature of the electrolyte must not be allowed to rise. If the  $D_2$  product gas is to overcome the pressure drop due to narrow tubes and a liquid head in the attached purification apparatus or reactor, the pressure in the cell must be maintained at a higher level by means of a throttling



Fig. 95. Electrolysis of D<sub>2</sub>O.

value in the  $O_2$  outlet. The  $D_2$  product still contains small amounts of  $O_2$  and  $D_2O$  vapor. Very pure gas may be obtained by heating the electrolysis product over platinized asbestos, followed by drying with liquid nitrogen. At 5 amp., two liters of  $D_2$  per hour is obtained.

Small quantities of  $D_2$  are stored in sealed glass flasks or over mercury. Distilled water can also be used as a sealing liquid. Larger amounts may be condensed in a metal flask cooled with liquid  $H_2$ . The liquid is then heated and thus forced through metal tubing into small steel cylinders.

Other equipment for electrolysis of  $D_2O$ , some of which is applicable to small-scale operation, is described by: F. Norling, Physik. Z. <u>36</u>, 711 (1935); C. M. Slack and L. F. Ehrke, Rev. Sci. Instruments (N.S.) <u>8</u>, 39 (1937); A. Sieverts and W. Danz, Z. Phys. Chem. B <u>38</u>, 46 (1937); M. M. Winn, J. Sci. Instruments <u>28</u>, 152 (1951); J. T. Lloyd, J. Sci. Instruments <u>29</u>, 164 (1952); R. W. Waniek, Rev. Sci. Instruments <u>21</u>, 262 (1950).

V. Other preparative methods: Reduction of  $D_2O$  with Fe or W at high temperatures.

SYNONYM:

Heavy hydrogen.

**PROPERTIES:** 

Colorless, odorless gas. Chemical properties analogous to  $H_2$ , but somewhat less reactive. In the absence of catalysts, mixtures of  $D_2$  and  $H_2$  are stable to about 500°C. In addition, no exchange with  $H_2O$  occurs at room temperature. M.p. -254.6°C, b.p. -249.7°C; d (liq., -253.1°C) 0.171. Very slightly soluble in water and other liquids.

REFERENCES:

- I. G. N. Lewis and W. T. Hanson. J. Amer. Chem. Soc. <u>56</u>, 1687 (1934).
- II. J. W. Knowlton and F. D. Rossini. J. Res. Nat. Bur. Standards 19, 605 (1937); unpublished experiments of G. Brauer.
- III. F. H. Spedding, A. S. Newton, J. C. Warf, O. Johnson, R. W. Nottorf, I. B. Johns and A. H. Daane, Nucleonics 4, 4 (1949).
- IV. C. L. Wilson and A. W. Wylie. J. Chem. Soc. (London) 1941, 596.
  - V. E. Zintl and A. Harder. Z. phys. Chem. B 28, 480 (1935);
     A. Farkas and L. Farkas. Proc. Roy. Soc. London <u>144</u>, 469 (1934).

## Hydrogen Deuteride

HD

$LiAlH_4$	$+ 4 D_2 O$	= LiOD	$+ Al(OD)_3$	+4 HD
37.94	80.12	24.96	81.02	12.09

This reaction is conducted in a 250-ml. two-neck flask provided with a reflux condenser and a magnetic stirrer. The other neck of the flask is closed with a rubber cap. The reflux condenser is connected to a receiver and a diffusion pump via cold traps, where the entrained liquid is condensed. Gas inlet lines with stopcocks allow each part of the apparatus to be evacuated separately or, if desired, to be filled with air or  $N_{2}$ .

About 150 ml. of n-butyl ether, dried over Na, is distilled into the reaction flask and 5.75 g. of  $LiAlH_4$  (40% excess) is then added under a nitrogen blanket. The mixture is frozen with liquid N<sub>2</sub>. The apparatus is then evacuated, and the flask contents are brought to boiling by careful heating. After 1.5 hours, it is again cooled with liquid  $N_2$ , the evacuation is repeated, and 5 ml. of 99.5%  $D_2O$  (see above,  $D_2O$ ) is added to the solidified mixture, using a hypodermic syringe to pierce the rubber cap. The gas evolution is started by melting the mixture and agitating with the magnetic stirrer. Because of the low reaction temperature, the flask becomes coated with ice on the outside. By repeated immersion in liquid  $N_2$ , the temperature is controlled so that the ice on the outer wall of the flask does not melt. As soon as the reaction subsides somewhat, two additional portions of 6.5 ml. of  $D_2O$  each are added (for a total of 18 ml. or 150% excess). The yield is 10 liters of HD. The purity is 97-99%.

### PROPERTIES:

Colorless, odorless gas. B.p.  $-251.02^{\circ}$ C; triplepoint  $-256.55^{\circ}$ C (92.8 mm.).

REFERENCES:

- A. Fookson, P. Pomerantz and E. H. Rich. J. Res. Nat. Bur. Standards 47, 31 (1951); Science (New York) 112, 748 (1950).
- J. Wender, R. A. Friedel and M. Orchin. J. Amer. Chem. Soc. 71, 1140 (1949).
- R. B. Scott and F. G. Brickwedde. Phys. Rev. (2) <u>48</u>, 483 (1935); <u>55</u>, 672 (1939).

## **Deuterium Fluoride**

### DF

I.

$D_2$	+ 2 AgF	$= 2  \mathrm{DF}$	+ 2 Ag
4.03	253.76	42.03	215.76

Some dry AgF is charged into a silver reaction flask provided with a manometer and an inlet tube that can be closed off. The AgF can also be produced by the action of  $F_2$  on the inner walls of the flask itself. After evacuation, pure, carefully dried  $D_2$ (see above,  $D_2$ ) is admitted into the flask. The latter is then closed and heated to  $110^{\circ}$ C until the pressure ceases to change. The DF formed is frozen out of the reaction mixture by cooling with liquid nitrogen, and excess  $D_2$  is drawn off by suction after opening the flask. The product is purified by high-vacuum distillation in which all connections and receivers must be of Ag or Cu.

To date, this method has been used only for producing small amounts of DF. Deuterium fluoride may be stored in vessels made of platinum, silver or copper. П.

$$2 C_{6}H_{5}COF + D_{2}O = (C_{6}H_{5}CO)_{2}O + 2 DF$$
248.22 20.03 226.22 42.03

The reaction is carried out in the apparatus shown in Fig. 96. The latter is flushed out with dry N  $_2$ , and atmospheric moisture is strictly excluded.

Silver flask  $\alpha$  is charged with 168 g. (1.5 moles) of benzoyl fluoride and chilled with Dry Ice-acetone freezing mixture. Then 5 g. (0.25 mole) of 99.5% D<sub>2</sub>O is added all at once under N<sub>2</sub>. The flask is then attached to the silver distillation apparatus. Brine at  $-15^{\circ}$ C is circulated through the condenser C, and the quartz receiver is cooled with Dry Ice-acetone to  $-80^{\circ}$ C. Cooling of



Fig. 96. Preparation of deuterium fluoride. a) Silver flask; b) thermometer well; c) jacketed glass condenser; d) paraffin-coated stopper; e) calcium chloride tube; f) quartz receiver.

flask a is then ceased and the latter is slowly heated to room temperature: the evolving DF is then distilled on a water bath at 80-90°C. To achieve analytical purity and separate entrained benzoyl fluoride, the distillation is repeated twice. The yield is 9.7 g. of DF (92% of theoretical).

III. Larger quantities of DF can be prepared by synthesis from the elements according to a method described by H. von Wartenberg for the production of HF; however, this requires extensive equipment.

IV. Aqueous solutions of deuterated hydrofluoric acid can be prepared by the condensation of DF in  $D_2O$  or by the reaction of very pure  $CaF_2$  with  $D_2SO_4$  (see also preparation of pure hydrofluoric acid, p. 145 ff.).

128

#### PROPERTIES:

Formula weight 21.01. Colorless, waterlike liquid; pungent odor; fumes in moist air. The vapors are very toxic. Chemical properties analogous to HF. The deuterium is exchanged for hydrogen in the presence of H<sup>+</sup>. B.p. +18.6°C. Very readily soluble in water.

REFERENCES:

- I. W. H. Clausen and J. H. Hildebrand. J. Amer. Chem. Soc. 56, 1820 (1934).
- II. G. Olah and S. Kuhn. Z. anorg. allg. Chem. 287, 282 (1956).
- III. H. von Wartenberg and O. Fitzner. Z. anorg. allg. Chem. <u>151</u>, 313 (1926).

## Deuterium Chloride

### DCl

I.

 $\begin{array}{ccc} 2 \ C_6 H_5 \text{COCl} + \ D_2 \text{O} = (C_6 H_5 \text{CO})_2 \text{O} + 2 \ \text{DCl} \\ 281.13 & 20.03 & 226.22 & 74.94 \end{array}$ 

The apparatus shown in Fig. 97 may be enlarged if larger amounts of DCl are desired. The long capillary tube from dropping funnel t, which reaches into the reaction flask r through the condenser k, ensures uniform addition of D<sub>2</sub>O to the benzoyl chloride in the flask in spite of small fluctuations of pressure during the reaction. In order to trap any benzoyl chloride entrained through the condenser by the DCl gas, trap f is cooled in an ice bath. Manometer m (with one arm open to the air) serves both as a safety valve and as a means for following the course of the reaction (if the outlet tube is closed off, the manometer will show whether the gas continues to evolve).

As an example of DCl preparation, 5 ml. of 99.6%  $D_2O$  is allowed to react with 210 g. of benzoyl chloride (2-3 molar excess) containing some porous boiling chips. At first, only a few drops of  $D_2O$  are added, while the mixture is carefully heated. This is continued until a moderate gas stream is developed. This temperature is maintained until all of the  $D_2O$  is added. By varying the heat input, gas formation can easily be regulated. As the flow decreases, the temperature is slowly increased to the boiling point of benzoyl chloride (197°C) and kept there until no further gas is evolved. At the end of the reaction, a stream of dry air is slowly introduced into the apparatus through the dropping funnel, without interrupting the refluxing, to expel all the DCl. The product is analytically pure and the yield is almost quantitative.



Fig. 97. Preparation of DCl from heavy water and benzoyl chloride. f) Condensation trap; k) reflux condenser; m) open arm manometer; r) reaction flask; t) dropping funnel

with capillary stem.

II.  $SiCl_4 + 2D_2O = 4DCl + SiO_2$ 169,89 40.06 149,88

Two thin-wall, vacuum-sealed ampoules containing 18 g. of carefully purified SiCl<sub>4</sub> and 1.8 g. of  $D_2O$  are shattered by shaking in an evacuated five-liter flask provided with a glass stopper with a stopcock sealed in. After 24 hours the flask is sealed to a high-vacuum system and the crude gas condensed in a liquid-nitrogen-cooled trap. The product may be further purified as in I or, even better, with a lowtemperature distillation column (see original literature for details).

Liquid deuterium chloride can be stored at low temperature. The gas may be stored in a sealed glass flask or over mercury.

Other preparative methods:

III. Reaction of anhydrous  $MgCl_{2}$  with  $D_{2}O$  at  $600^{\circ}C$ :

 $MgCl_2 + D_2O = 2 DCl + MgO.$ 

Yields very pure DCl on distillation. IV. Reaction of very pure NaCl with  $D_2SO_4$ .

V. Aqueous solutions of heavy hydrochloric acid are prepared by condensation of DCl in  $D_2O_{\bullet}$ 

SYNONYM:

Heavy hydrogen chloride.

PROPERTIES:

Formula weight 37.47. Chemical properties analogous to HCl. In the absence of moisture and catalysts, no deuterium exchange occurs in gaseous mixtures of HCl and DCl. However, an exchange reaction occurs instantaneously in solvents containing  $H^+$ . M.p. -114.8°C, b.p. -81.6°C,  $t_{\rm Cr}$  +50.3°C.

**REFERENCES:** 

- I. H. C. Brown and C. Groot. J. Amer. Chem. Soc. 64, 2223 (1942).
- II. K. Clausius and G. Wolf. Z. Naturforsch. 2a, 495 (1947).

- III. G. N. Lewis, R. T. Macdonald and P. W. Schutz. J. Amer. Chem. Soc. 56, 494 (1934).
- IV. A. Smits, G. J. Muller and F. A. Kröger. Z. phys. Chem. B38, 177 (1937); see also O. E. Frivold, O. Hassel and S. Rustad. Phys. Z. 38, 191 (1937).

## Deuterium Bromide

### DBr

I.

 $D_2 + Br_2 = 2 DBr$ 4.03 159.83 163.86

The glass apparatus shown in Fig. 98 is used. Prior to the run, it is evacuated for a considerable time via P. Flask C is charged with carefully purified  $\operatorname{Br}_2$  from D (see section on  $\operatorname{Br}_2$ ) by moving the plug S of the dropping funnel (which has no stopcock). The flask is then heated to 48°C. Dry  $D_2$  (see p. 121) enters at A at a rate of about two liters/hour and passes through stopcock B(lubricated with phosphoric acid-graphite and sealed with mercury) into C, where it mixes with the bromine vapor which is replenished during the reaction from the dropping funnel. The  $D_2$ -Br<sub>2</sub> mixture flows into the Vycor combustion tube R, which is filled with



Fig. 98. Preparation of DBr. B) Hg seal stopcock, lubricated with graphite-phosphoric acid; D) bromine storage vessel; C) reaction flask; R) Vycor tube filled with porcelain chips; E, F) condensation traps; K) column with Cu turnings.

small pieces of porcelain and wound with heating wire so that the front part is heated to  $80^{\circ}$ C and the back to  $700^{\circ}$ C. Here, 99%of the D<sub>2</sub> is converted to DBr. Excess Br<sub>2</sub> is separated in trap E, kept at -40°C, and in column K filled with clean copper turnings. The DBr, condensed in the liquid nitrogen-cooled receiver F, can be purified several times by repeated fractional distillation in high vacuum (see purification of DI, p. 133, as well as method II). The yield is almost quantitative.

Π.

 $\begin{array}{rrr} PBr_{3} + 3 D_{2}O = 3 DBr + D_{3}PO_{3} \\ 270.73 & 60.09 & 245.79 & 85.02 \end{array}$ 

The vacuum-sealed ampoules with the starting materials are broken by vigorous shaking in an evacuated 5-liter flask closed by a ground glass stopper with a stopcock. To complete the deuterolysis, the mixture is left standing in the dark for two days, with occasional shaking. The reaction should not be accelerated by heating or disproportionation (4  $D_3PO_3 = 3 D_3PO_4 + PD_3$ ) will occur, and the DBr will be contaminated with PD<sub>3</sub>. After sealing the flask to a high vacuum system, the impure gas is condensed in a receiver cooled with liquid nitrogen and purified by fractional distillation, using a low-temperature distillation flask (see Part I, p. 69).

Unlike method I, only half of the deuterium feed is converted to desired products.

Deuterium bromide is stored either as a liquid at a low temperature or as a gas in a sealed glass flask. Pure DBr reacts with Hg only on long exposure.

III. Aqueous solutions of heavy hydrobromic acid can be obtained by condensation of DBr in  $D_2O$ .

SYNONYM:

Heavy hydrogen bromide.

PROPERTIES:

Formula weight 81.93. Chemical properties analogous to those of HBr. In the presence of H<sup>+</sup>, exchange occurs. M.p.  $-87.5^{\circ}$ C, b.p.  $-67^{\circ}$ C; t<sub>Cr</sub> +88.8°C.

**REFERENCES:** 

- I. C. L. Wilson and A. W. Wylie. J. Chem. Soc. (London) <u>1941</u>, 596.
- II. K. Clusius and G. Wolf. Z. Naturforsch. 2a, 495 (1947).

132

# Deuterium lodide DI

I.

 $D_2 + I_2 = 2 DI$ 4.03 253.84 257.87

The glass apparatus of Fig. 99, in which all the joints are fused, is used. The 5-liter flask A contains some platinum sponge or platinized asbestos (see section on Platinum Metals) which is initially calcined for a few hours in high vacuum (evacuate through C) at 450°C. Dry, H<sub>2</sub>-free air (in order to prevent adsorption of light hydrogen on the platinum) is allowed to flow in and 35 g. of carefully purified iodine (see that section) is added to the flask through C. Evacuation is then resumed until all of the air is displaced by I<sub>2</sub> vapor. Pure D<sub>2</sub> (see above) is then introduced by means of a Toepler pump, until a pressure of 120 mm. is reached. The system is then melt-sealed at C. The flask is heated in an air bath for six hours at 370°C; over 90% of the D<sub>2</sub> is converted to DI. The impure gas is separated from the



Fig. 99. Preparation of DI. A) 5-liter flask with catalyst; B) break-seal valve; K) seal breaker; E, F) condensation traps; 1,2) sealing points.

unconverted starting materials by fractional distillation. For this purpose, the right part of the apparatus, separated from the reactor by the seal B, is evacuated with stopcocks H and D open. Then D is closed and the tip at B is broken by moving the glasscovered iron slug K with an electromagnet. Trap E is chilled with liquid nitrogen and D can then be opened. With stopcock H closed, the contents of A are distilled into E. Next, the system is melt-sealed at 1 and evacuated briefly through H, and distillation from E to F is repeated, whereupon E is warmed to  $-79^{\circ}$ C with a Dry Ice bath and  $\overline{F}$  is cooled with liquid nitrogen. Finally, the tube is sealed off at 2. The DI contained in  $\overline{F}$  is pure white, i.e., completely free from elemental iodine. Deuterium iodide can only be stored in condensed form at low temperature.

### Other preparative methods:

II.  $P + 5I + 4 D_2O = 5 DI + D_3PO_4$ . The readily formed side products  $PD_3$  and  $PD_4I$  contaminate the DI. The method utilizes only about half of the deuterium introduced.

III. Solutions of heavy hydriodic acid are obtained by reaction of  $D_2S$  with iodine in the presence of  $D_2O$ :  $D_2S + I_2 = 2 DI + S$ .

Deuterium sulfide is introduced with shaking into an icecooled suspension of  $I_2$  in  $D_2O$  placed in a closed recirculating glass apparatus with all joints melt-sealed. Unconverted  $D_2S$ is reintroduced into the reaction mixture. The heavy hydriodic acid formed is separated from the precipitated sulfur by filtration (in the absence of air) and separated from the dissolved  $D_2S$  by prolonged evacuation.

#### **PROPERTIES:**

Formula weight 128.93. Chemical properties analogous to HI. Deuterium is replaced by hydrogen in the presence of  $H^+$ . M.p. -52.0°C, b.p. -36.2°C; t<sub>cr</sub> +148.6°C.

REFERENCES:

- I. D. Rittenberg and H. C. Urey. J. Amer. Chem. Soc. <u>56</u>, 1885 (1934); J. R. Bates, J. O. Halford and L. C. Anderson. J. Chem. Phys. 3, 415 (1935).
- II. K. Clusius and G. Wolf, Z. Naturforsch. 2a, 495 (1947). III. H. Erlenmeyer and H. Gärtner. Helv. Chim. Acta 19, 146 (1936).

### Deuterium Sulfide

### D₂S

I.

 $\begin{array}{r} Al_2S_3 + 6 D_2O = 3 D_2S + 2 Al(OD)_3 \\ 150.12 & 120.18 & 108.27 & 162.03 \end{array}$ 

To prepare  $Al_2S_3$  (see the section on Aluminum), a stoichiometric mixture of C.P. Al powder and C.P. S is placed in a sulfur-lined Hessian clay crucible. The commercial Al powder used must be washed several times with pure benzene to remove all oils; it is then heated for some time to  $150^{\circ}$ C in high vacuum. The reaction mixture is ignited with the help of a Mg strip (cautionvery violent reaction!) and the crucible is covered. The  $Al_2S_3$ product is crushed while still hot, placed in ampoules, and degassed for several hours in high vacuum at 150 to  $180^{\circ}$ C. The ampoules are sealed in vacuum. The  $D_2O$  reagent is also placed in small ampoules and carefully degassed in vacuum, and the ampoules are sealed.

To make D<sub>2</sub>S, 20 g, of Al<sub>2</sub>S<sub>3</sub> and 7 g, of D<sub>2</sub>O in sealed ampoules (the excess  $Al_2S_3$  is an excellent drying agent for the product gas) are placed in a 5-liter flask with a ground glass stopper provided with a stopcock. After evacuation to about  $10^{-4}$  mm., the stopcock is closed and the connection to the vacuum source is sealed off. The ampoules are broken by shaking the flask, starting evolution of the gas. Heavy water vapor, which may condense in the uppper part of the flask, is made to react by warming the walls or by coating them with unreacted Al<sub>2</sub>S<sub>3</sub>. The flask is left standing in the dark, with occasional shaking, for about one week. After this, it is sealed to a vacuum system provided with several traps for fractional condensation (see Part I, p. 67 f.). Small amounts of D<sub>2</sub> are separated from the impure gas by condensing with liquid nitrogen and fractionating by repeated slow distillation (bath liquids: Dry Ice mixture and liquid nitrogen). The D<sub>2</sub>S is then pure enough not to attack metallic Hg even after several weeks of contact. The yield is somewhat lower than stoichiometric.

The product may be stored in condensed form at low temperature or as a gas over dry paraffin oil.

II. Other preparative methods: Decomposition of CaS with  $D_2O$  in the presence of  $MgCl_2$ .

### PROPERTIES:

Formula weight 36.10. Chemical properties analogous to  $H_2S$ . In solvents containing H<sup>+</sup>, deuterium is replaced with hydrogen. M.p. -86°C, t<sub>cr</sub> +99.1°C.

REFERENCES:

 I. M. Fonzès-Diacon. Comptes Rendus Hebd. Séances Acad. Sci. <u>130</u>, 1314 (1900); preparation of Al<sub>2</sub>S<sub>3</sub>: A. Kruis and K. Clusius. Z. phys. Chem. <u>B38</u>, 156 (1937); see also H. Erlenmeyer and H. Gärtner. Helv. Chim. Acta <u>19</u>, 146 (1936); O. E. Frivold, O. Hassel, et al. Phys. Z. <u>39</u>, 224 (1938); <u>38</u>, 191 (1937).
 II. T. Larsén. Z. Phys. 111, 391 (1938).

# Deuterosulfuric Acid

# $D_2SO_4$

 $D_2O + SO_3 = D_2SO_4$ 20.03 80.06 100.09

All joints of the glass apparatus, shown schematically in Fig. 100, are fused. Two ampoules,  $F_1$  and  $F_2$ , contain SO<sub>3</sub>, carefully pre-

purified by sublimation. The ampoules are placed in glass vessels  $A_1$  and  $A_2$  on top of sealed-in test tubes  $S_1$  and  $S_2$  so that their tips are directly below the glass-covered slugs  $K_2$  or  $K_3$ . This placement of ampoules on "stems" eases the job of the glass blower who must fuse the joints of the apparatus. After evacuation



Fig. 100. Preparation of  $D_2S$ .  $F_1$ ,  $F_2$ ) ampoules containing  $SO_3$ ;  $A_1$ ,  $A_2$ ) containers for  $SO_3$  ampoules;  $K_1$ ,  $K_2$ ,  $K_3$ ) seal breakers;  $M_1$ ,  $M_2$ ,  $M_3$ ) electromagnets; H) break-seal value; C, D, E) receivers (can be cooled); G, L) sealing points.

and sealing off at  $P_1$ , ampoules  $F_1$  and  $F_2$  are broken by manipulation of  $K_2$  and  $K_3$  with the electromagnets  $M_2$  and  $M_3$ . The SO<sub>3</sub> passes through two U tubes filled with P<sub>2</sub>O<sub>5</sub>-glass wool and is condensed in C by means of a Dry Ice bath. During this sublimation the left part of the apparatus, which is separated by the glass barrier  $\overline{H}$ , is filled with D<sub>2</sub>O through G and that inlet sealed, and the  $D_2O$  is frozen in E by means of a Dry Ice bath. The system is then evacuated at  $P_2$  and sealed off by fusion. After the sublimition of the  $SO_3$ , the tubing is also fused at L. The barrier H is now broken with the glass-encased iron slug  $K_1$  moved by electromagnet  $M_1$ , with the glass splinters falling into receiver D. The SO<sub>3</sub> is then condensed on the D<sub>2</sub>O in E by slow heating of C. After careful melting of the reaction mixture, deuterosulfuric acid of any desired concentration is obtained in E. The concentration is regulated by the proportion of D<sub>2</sub>O and SO<sub>3</sub>. The yield is quantitative, based on  $D_2O_2$ .

The product is stored in glass vessels.

PROPERTIES:

Colorless liquid, with an oily consistency. Chemical properties analogous to  $H_2SO_4$ ; the deuterium is ionic and exchangable with light hydrogen. This should be kept in mind when mixing with solvents containing H<sup>+</sup>. Miscible with water in all proportions.

REFERENCE:

F. Fehér. Ber. dtsch. chem. Ges. 72, 1789 (1939).

### Deuteroammonia

### ND<sub>3</sub>

I.

 $\begin{array}{rrr} Mg_{3}N + 6 D_{2}O = 2 ND_{3} + 3 Mg(OD)_{2} \\ 100.98 & 120.18 & 40.11 & 181.05 \end{array}$ 

The one-piece glass apparatus shown in Fig. 101 is used. The three U tubes are filled with 30 g. of  $Mg_3N_2$  (see section on Magnesium for preparation), sealed to each other, and degassed at 400°C for some time in high vacuum. Meanwhile, flask *P*, separated from the rest of the apparatus by the glass wall *D*,



Fig. 101. Preparation of ND<sub>3</sub>. P, Q, R, T) receivers (can be cooled); D) break-seal valve; K) seal breaker; M) electromagnet.

is filled with 7 g. of  $D_2O$  which is freed of air by repeated freezing and melting in high vacuum. An excess of Mg<sub>3</sub>N<sub>2</sub> will thoroughly dry the product gas. After both vacuum connections are fused, Q is cooled with liquid nitrogen and barrier D is broken by moving the glass-encased iron slug K with the electromagnet. Reaction between  $D_2O$  and Mg<sub>3</sub>N<sub>2</sub> starts immediately, and the ND<sub>3</sub> formed condenses in Q. When the  $D_2O$  from P is completely evaporated, the U tubes are heated for some time by fanning with a flame. The product collected in Q is sublimed twice in high vacuum to free it from  $D_2O$ . To accomplish this, R is cooled with liquid nitrogen and Q is warmed in a Dry Ice bath to  $-78^{\circ}$ C. Then the connection between Q and R is sealed off, R is warmed to  $-78^{\circ}$ C, and T is cooled with liquid nitrogen. The yield is almost quantitative, based on  $D_2$ O.

The product may be stored in condensed form at low temperatures or as a gas over Hg.

II. Aqueous solutions of heavy ammonia are prepared by condensing  $ND_3$  in  $D_2O$  in high vacuum.

PROPERTIES:

Formula weight 20.05. Chemical properties analogous to  $NH_3$ . In the presence of solvents containing H<sup>+</sup>, deuterium is replaced by hydrogen. M.p. -73.6°C, b.p. -31.1°C, t<sub>cr</sub> +132.3°C.

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# Deuterophosphoric Acid

# D<sub>3</sub>PO<sub>4</sub>

 $P_2O_5 + 3D_2O = 2D_3PO_4$ 141.95 60.09 202.04

In the Simon and Schulze method, heavy phosphoric acid is prepared via the gas-phase reaction of pure  $D_2O$  and  $P_2O_5$  in vacuum, using an apparatus consisting of flasks connected with



Fig. 102. Preparation of deuterophosphoric acid solution.

138

ground joints. The approximately  $53\% D_3PO_4$  solution formed in the process is refluxed for 6.5 hours to produce orthophosphoric acid. Atmospheric moisture is blocked by a  $P_2O_5$  tube on the condenser. The escaping  $D_2O$  is recovered by freezing it out in a trap cooled with a Dry Ice-ether bath, which is inserted between the condenser and the  $P_2O_5$  tube.

The condenser is then replaced with an adapter provided with an inlet and an outlet tube (see Fig. 102) and  $O_3$  is passed through the acid solution for one hour. During this treatment, the flask is kept in warm, 70°C water. Most of the  $D_2O$  vapor produced is retained in the flask by chilling the neck with a condenser coil. Small amounts of vapor which escape are condensed in a trap. Two tubes filled with  $P_2O_5$  shield against atmospheric moisture.

The purified  $53\% D_3 PO_4$  solution is concentrated to 83% at  $45^{\circ}C$  in the apparatus shown in Fig. 195 (p. 543). No pyro acid should be produced at this temperature. The course of the concentration is observed by weighing the cooling trap. Further concentration by evaporation is impossible because partial conversion to the pyro acid occurs.

The purity of the acid thus prepared may be ascertained from a pure yellow precipitate of silver phosphate, which does not discolor even upon boiling.

**REFERENCE:** 

A. Simon and G. Schulze. Z. anorg. allg. Chem. 242, 326 (1939).

# SECTION 2

# Hydrogen Peroxide

## M. SCHMEISSER

# Hydrogen Peroxide

## H<sub>2</sub>O<sub>2</sub>

Staedel [1] was the first to describe a method, later altered in various ways by others [2-6], according to which 30% H<sub>2</sub>O<sub>2</sub> solution is distilled to remove most of the water, the residual H<sub>2</sub>O<sub>2</sub> is crystallized by cooling, and the crystals are separated from the mother liquor.

A 500-ml. distilling flask is provided with a standard male ground joint, onto which is placed a female glass cap, equipped with a distillation capillary. The side tube of the capillary is connected with ground joints to a spiral condenser, which empties into a receiver of about 200-ml. capacity. After the introduction of 180 ml. of Perhydrol (stored in bottles coated with paraffin wax), the flask is placed on a water bath (45 to  $50^{\circ}$ C) and the material is distilled over a period of about 3.5 hours at a pressure of 16 to 22 mm. Thus, about 150-160 ml. of water and some H<sub>2</sub>O<sub>2</sub> are removed. The residue contains approximately 98% H<sub>2</sub>O<sub>2</sub>. The volume of water to be distilled may be marked off on the previously tared receiving flask. (If the temperature of the water bath rises above 52°C, the concentrated H<sub>2</sub>O<sub>2</sub> turns yellow and should be discarded.) The concentrated product may be removed from the flask without any danger of decomposition. (If a female ground joint were to be used at the neck of the flask, the decomposition on the rough surface would be appreciable.)

Further work-up to obtain 100%  $H_2O_2$  is carried out in the following manner: A short, large-diameter test tube of 25-30 ml. capacity, coated inside with paraffin or ozokerite wax, is half filled with concentrated  $H_2O_2$ , closed off with a paraffin-coated rubber stopper, and cooled at  $-35^{\circ}C$  for half an hour. Meanwhile, seed crystals are prepared by freezing about 1 ml. of the same  $H_2O_2$  in liquid nitrogen. (For the melting-point diagram of the system  $H_2O_2-H_2O$ , see [7].) Colorless, needle-shaped crystals form immediately after seeding. After waiting for about a minute, the crystals are quickly transferred to a precooled  $(-30^{\circ}C)$  centrifuge tube such as that shown in Fig. 103. Following a brief centrifugation (either by hand or in a manually operated centrifuge) the crystals are transferred into another large-diameter test tube and remelted.

In order to shorten the melting process, the tube with the  $H_2O_2$  is placed in a beaker of warm water (30°C). When all the crystals have melted, the peroxide is again cooled to  $-35^{\circ}$ C. On standing in the cold bath for about 10 minutes, the colorless, needle-shaped crystals re-form, usually spontaneously, and are again immediately separated from the mother liquor by centrifuging in an identical tube. If crystallization does not occur spontaneously, seeding is repeated.

The resultant crystals decompose very easily at room temperature, liberating  $O_2$ . They are therefore stored in closed paraffin-coated containers, which must be kept cold. Fig. 103. Centrifuge tube for pure hydrogen peroxide.

The aqueous  $H_2O_2$  solutions separated by centrifuging may be reconcentrated by distillation.

A single crystallization of the approximately 98% product in the test tube, followed by separation of the mother liquor, yields a product which is only about 99% pure.

According to Hurd and Puterbaugh [8], 80-90% H<sub>2</sub>O<sub>2</sub> starting material can also be obtained by mixing 30% H<sub>2</sub>O<sub>2</sub> solution with twice its amount of p-cymene, followed by distillation of the mixture at 50°C, using an aspirator vacuum. Most of the water and p-cymene is thus removed. After mechanical separation of the remaining p-cymene-H<sub>2</sub>O<sub>2</sub> mixture, further processing is carried out as described above.

Additional preparative methods: For the preparation of  $H_2O_2$  of spectroscopic purity, the process reported by Fehér [9] for  $D_2O_2$  can be referred to. This process is based on the work of Pietzsch and Adolph [10] and involves the reaction of persulfate with steam.

A method for the production of single crystals of  $H_2O_2$  has been described by Fehér [11].

PROPERTIES:

Formula weight 34.016. M.p.  $-1.7^{\circ}$ C, b.p. (extrapolated) 157.8°C; d(liq.) (0°C) 1.46, d (solid) 1.64.

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# SECTION 3

# Fluorine, Hydrogen Fluoride

# H. von WARTENBERG

### Fluorine

### $\mathbf{F}_2$

Fluorine is produced at present either by electrolysis of molten KHF<sub>2</sub>, being liberated at a graphite anode between 200 and 300°C (slight contamination by  $CF_4$ ), or from molten  $KF \cdot 3$  HF at a Ni anode at approximately 100°C, according to the method of Lebeau (if the melt contains water, contamination by  $O_2$ ). The latter method is the most tractable and has been tested extensively. The only suitable vessel materials are Fe (which develops a rust coating on the surface), Cu and Mg. The salt is available in a pot made entirely of brazed Cu with about 2-mm. wall thickness (Fig. 104). The cover rests lightly on three pieces of CaF, placed in the upper trough, and held in place by a layer of cement. Alternatively, the cover rests on a gasket cut from 5-mm.-thick soft rubber sheet, which in turn rests on the flat lip of the pot. To slow down the penetration of humid air into the trough, the latter is also packed with CaF<sub>2</sub> powder. The anode, made of 3-mm. nickel wire, is attacked only at its extreme end. Therefore, greater service life is obtained by coiling the electrode or attaching to it a 1-cm. nickel rod. Thus, the useful life of an electrode may be extended to the electrolysis of approximately two complete salt batches. After that, a new anode can easily be inserted into the connecting copper adapter. The melt usually spatters and creeps up the walls. This is the reason for placing the insulation so high in the thick-wall Cu adapter, which is cemented into the upper tube with litharge-glycerol. This adapter should preferably be taken out with strong pliers at the end of each electrolysis cycle and thoroughly washed. The sturdy fluorite stopper is secured with the same cement. It is even simpler to use a one-hole rubber stopper, coated with polytrifluorochloroethylene oil. The anode is surrounded by a copper or, better, a nickel tube. Three struts extend downward from the tube and are attached, by keyed copper connections, to a Cu plate. This plate acts as protection against the  $H_2$  rising from the wall of the pot. The pot serves as the cathode. This is the best arrangement since it leaves a large free cross section for the electrolyte salt. The Cu tube may become coated with an insulating layer of fluoride, which, however, sometimes dissolves if the temperature rises too high, making it necessary to replace the tube from time to time. The tube cannot be replaced with sintered alumina, which dissolves in the melt. This inconvenience is unfortunately encountered with all apparatus for production of fluorine. In order to generate the  $F_2$  under a



Fig. 104. Preparaation of fluorine.

pressure of about 10 cm. of water, the pot is intentionally made tall. The outlet tubing must be at least 6 mm, inside diameter, since some electrolyte particles are always entrained with the gas. A single electrolysis batch requires 1.2 kg. of difluoride of the highest purity and 300g. of freshly distilled, anhydrous HF. The apparatus is put on a hot plate which is placed on an ordinary platform scale, making it easy to determine from the loss in weight (300 g.) whether refilling is necessary and to avoid unnecessary overshooting of the temperature. At the beginning of the preparation a horizontally directed Bunsen flame is used as auxiliary heat, so that the salt is melted and heated to 70°C in half an hour (attach a thermometer). No heat is applied during the electrolysis, which proceeds at about ten volts and about 4-5 amp. (with Ni rod anodes, 6 amp. yields

40 ml. of  $F_2/min$ .). The hot plate serves only to keep the contents of the pot in the liquid state during interruptions in the run. After electrolyte depletion, the cover is removed and the residual salt is allowed to solidify while the container is being rotated, thus creating a cavity (work under a hood, wearing goggles;  $p_{HF}$  at 150°C is 130 mm.). Fresh HF is added to the cavity, the vessel is covered with a piece of Cu sheet, and the HF is left to be absorbed by the salt overnight. After heating to 90°C, the cover with the anode can be replaced. In order to remove the small amounts of HF, the product gas is first led through a  $10 \times 50$  mm., Dry Icecooled Cu bottle brazed to a tube of poorly heat-conducting nickelsilver alloy ( $p_{HF} \sim 1$  mm.) and then through a Cu tube filled with granular NaF and provided with copper plug valves. Detection of F<sub>2</sub> in the exit stream (as well as at leaks) is easy. A jet of illuminating gas or a rag soaked in machine oil and attached to a wire will ignite on contact with F<sub>2</sub>. If the salt mass has been well dried, the traces of water disappear completely after the first hour of electrolysis. Any O2 which may be formed cannot be separated from the  $F_2$ ; it may be determined, although not conveniently, by shaking with Hg [2].

With the packing materials now on the market, for example, Teflon, it is easy to build similar equipment made of somewhat thicker sheet copper or of cast magnesium by using Teflon-insulated gaskets and tightening the apparatus with screws. When a Cu pot with 5-mm.-thick walls, 35 cm. high and 15 cm. in diameter, containing 5 kg. of  $KHF_2$ , is used, a current of 5 amp. may be applied. The HF gas can then be fed almost continuously to the outer chamber, thus replacing the raw material as it is consumed.

### PROPERTIES:

Atomic weight 19.00. M.p.  $-223^{\circ}$ C, b.p.  $-187^{\circ}$ C; d (liq.)1.11, d (gas)1.31 (air = 1). Fluorine does not attack quartz and very dry glass. For heating in a F<sub>2</sub> atmosphere, Pt tubes are used, or even better, sintered alumina tubes (up to 600°C), while Cu tubes are useful up to 350°C and Ni to 600 or 700°C. Teflon or Kel-F is used as gasketing material and Kel-F grease is used for lubrication of stopcocks and ground joints.

For commercial apparatus, see [3, 4].

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## Hydrogen Fluoride

### $\mathbf{HF}$

**Precautionary measures**. Since hydrogen fluoride solutions, particularly when concentrated, cause extremely painful and protracted burns, a paste made of magnesium oxide with a little glycerol should be kept on hand when working with larger quantities. The eyes must be protected and rubber gloves must be worn.

## I. CRUDE HYDROFLUORIC ACID [1]

Laboratory preparation of this material will almost never be undertaken. To obtain 0.25 kg. of HF, 1 kg. of finely pulverized fluorite or, even better, cryolite (for Si-free HF) is vigorously heated with 2.25 to 2.5 kg. of 97.5% As-free  $H_2SO_4$  in a small autoclave placed on an air bath. A lead tube 1.5 meters long and 2 cm, in diameter, with an attached Liebig condenser, is soldered to the cover. Run duration, 4 hours. The product is collected in a copper flask cooled with ice-salt mixture. Impurities: H<sub>2</sub>SiF<sub>6</sub>, HCl, H<sub>2</sub>SO<sub>3</sub>, H<sub>2</sub>SO<sub>4</sub>, HSO<sub>3</sub>F, Pb.

## II. PURE, 35% HYDROFLUORIC ACID

This solution is commercially available in polyethylene bottles and is already quite pure. For further purification, it is distilled from a NaF-containing Pt retort into a Pt receiver, leaving behind the  $SO_4^{2-}$  and  $SiF_6^{2-}$  ions. A little PbCO<sub>3</sub> is added to remove the Cl<sup>-</sup>; this yields PbClF, which is insoluble in concentrated HF. An excess of PbCO<sub>3</sub> does no harm, even in the presence of H<sub>2</sub>SO<sub>4</sub>. Organic material is removed only when KMnO<sub>4</sub> is added (dropwise).

Vessel materials: Pt, Ag, Cu, Mg (but not Pb), celluloid (may be easily shaped in warm water), polyethylene, paraffin, metal dishes coated with Bakelite, Teflon, etc. See also section on  $F_2$ . Boiling points of various  $H_2O/HF$  mixtures are given in [4].

### III. ANHYDROUS HYDROGEN FLUORIDE

Distillation of 1.2 kg. of anhydrous KHF<sub>2</sub> at 500°C yields 250 g. of HF. Technical grade KHF<sub>2</sub> is dissolved in warm water, some PbCO<sub>3</sub> is added to eliminate  $Cl^-$ , the K<sub>2</sub>SiF<sub>6</sub> and PbClF are allowed to settle, and the clear solution is evaporated in Cu or Mg dishes until crystallization occurs. Alternatively, a quantity of hydrofluoric acid is divided into two equal parts, one of which is neutralized with K<sub>2</sub>CO<sub>3</sub>, mixed with the other part and evaporated. Filter hot through a Cu funnel. The crystals which separate on cooling are allowed to drain and are dried initially on filter paper, at 100°C. Further drying must be carried out with great care at 130 to 140°C. A thin layer of crystals is placed on a Mg or Cu sheet turned up 1-2 cm. along the edges, and the sheet is placed on a large hot plate. After 2-3 days the crystals are ground in a coffee mill and dried for another day, after which they are stored in paraffincoated glass bottles. The dust from the salt is disagreeable. The compound can be obtained more readily by decomposition of NaHF2, but the decomposition starts already drying, and NaHF, is therefore not recommended as a raw material.

Distillation is carried out in a Cu flask, 24 cm. high, provided with a conical ground stopper, 4 cm. in diameter, held in place with screw clamps (Fig. 105). The cone is lubricated with graphiteparaffin oil or Kel-F grease to prevent freezing of the joint. The 1-m.-long Cu tube must be 2-2.5 cm. in diameter, since salt is carried over and inconvenient plugging can occur in the middle of

the run. The glass condenser is gasketed with a piece of rubber hose, and to slow down the decomposition, a few turns of watercooled tin tubing are wound on it. The Cu tube is allowed to stand in HCl/Br<sub>2</sub> until it is bright, or the initial HF fraction will be brown. The retort should be provided with a Cu thermowell, projecting upward for 5 cm. from the bottom and externally brazed in place, to accommodate a copper-constantan thermocouple. The retort is heated either on a multiple burner in an iron jacket or, preferably, in an electric furnace. Before filling it should be cleaned with HCl until bright and dried carefully in a stream of  $CO_2$ . The Cu receiver (250 ml.) is connected by a special coupling (Fig. 106) and must be cleaned with HCl until bright. It is then dried and reduced in a stream of H<sub>2</sub> at 300°C. During distillation, use an ice-salt mixture (-10°C) for cooling. A second container provides protection against overflow. The retort is first heated slowly for approximately 3/4 hour to about 400 to 500°C, that is until HF begins to drip into a Pt dish below (leave the screw coupling at E open). Vapor pressures of KF and HF are given in [6]. After about 10 ml. has been collected, a test is made to determine whether a strip of filter paper gelatinizes immediately. If so, the acid is anhydrous. Now the coupling at E is tightened (use pliers and rubber gloves) and the freezing bath is put in place. Raising of the temperature to 500°C must be accomplished with great care, since at that temperature KF begins to separate from the melt and violent evolution of HF also occurs [2]. The operation is decidedly more convenient if a thick-walled Cu capillary (shown with dotted lines) is brazed to the Cu tube. The capillary should dip about 2 cm. into a Hg pool in a Pt dish. If this is done, the couplings on the receivers may be tightened before the run and the first few milliliters allowed to drip from the capillary until the paper test shows the absence of water. The capillary may then be sealed off simply by raising the level of the mercury. Should the evolution of HF be too violent, excess HF can escape through the capillary, accompanied, of course, by noxious fumes.



Fig. 105. Distillation of anhydrous hydrofluoric acid.

A) distillation retort (Cu); B) condenser (inner tube of Cu); C) receiver (Cu),  $-10^{\circ}$ C; D) second receiver,  $-10^{\circ}$ C; E,F) conical screw couplings.

The evolution subsides after about 3 hours and the heating is discontinued. Loosen E and F and close them off with Cu cones. If the distilled acid is to be stored, the tubes can be closed off at E and F with solid Cu cones equipped with screw caps. After cooling, the contents of the retort may be dissolved in boiling water and regenerated with aqueous HF.



Fig. 106. Details of the conical copper coupling. The cones are interchangeable.

For regeneration, the solution is treated in a large Cu or Mg dish with sufficient pure, commercial (35-40%) HF to turn litmus completely red. The solution is then evaporated over an open flame but not to the point where spattering occurs. The salt mass is then crushed and dried as described previously.

The resultant HF still contains traces of entrained KF and can be redistilled at  $30-35^{\circ}$ C into the second receiver. For very pure HF, silver equipment should be used [2]. Since organic substances are immediately decomposed by HF vapor, joints can be made tight only by means of the Cu cones described. Alternatively fused sulfur may be used, with picein or a like substance covering cracks in the sulfur mass. As a temporary expedient, resulting in not quite anhydrous acid, the seal can be made with well-dried lithargeglycerol cement.

Liquid HF is available in steel tanks or cylinders. (Heat gently with flame to make it flow.)

In order to render the liquid HF completely anhydrous, fluorine may be bubbled through using a silver capillary. The  $F_2$  gas decomposes traces of water. Thus a cylinder HF can be partially dried by somewhat loosening the main valve with a wrench in a cold room and then putting the cylinder in a container of appropriate height, filled with ice-salt mixture; after the cylinder has cooled down, the valve is removed completely and  $F_2$  is bubbled through or half an hour from a fluorine cylinder through a silver capillary; the valve is then screwed back on and tightened with the wrench. The fluorine pressure produced in the usual laboratory generators is not sufficient for the HF cylinders. For all practical purposes,  $F_2$  does not dissolve in liquid HF. Instead of using fluorine, the water may be removed by dropwise addition of thionyl chloride, which liberates gaseous HCl, SO<sub>2</sub> and SOF<sub>2</sub>, all insoluble in HF [K. Wiechert, Z. anorg. Chem. 261, 314 (1950)]. Pure HF for very small scale experiments be easily obtained (together with hydrogen) by placing a copper boat with welldried  $PbF_2$  in a platinum or copper tube in front of the compound to be reacted with the nascent HF. The boat is heated to red heat in a stream of hydrogen.

Conduits for HF gas are made of well-dried copper tubes provided with conical copper joints. In such tubes HF gas may be heated to  $1000^{\circ}$ C (of course, protection from the atmosphere must be provided). Sintered alumina tubes may be used up to  $500^{\circ}$ C. Copper caps, temporarily cemented on with lithargeglycerol, may serve as closures, but it is better to solder them onto the alumina tube. To accomplish this, the alumina tube is electrolytically coated with copper, and soft solder is used for attaching the caps. Completely anhydrous HF does not attack quartz. Lead and organic substances are destroyed, except for polymerized tetrafluoroethylene (Teflon) [5]. Copper (not brass) stopcocks or, preferably, platinum valves (Bodenstein design) serve to shut off the flow. It is advisable to make the stopcock body of copper and to turn out the plug from a Teflon block on a lathe. Such a plug turns easily and needs no lubrication.

### PROPERTIES:

Formula weight 20.01. B.p. 19.5°C; for b.p. at various pressures see [3]; m.p.  $-85^{\circ}$ C; d (liq.) 0.987. t<sub>Cr</sub> 188°C, p<sub>Cr</sub> 66.2 kg./cm.<sup>2</sup>, d<sub>Cr</sub> 0.29.

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# SECTION 4

# Fluorine Compounds

W. KWASNIK

# General Remarks

In view of the special position occupied by fluorine among the halogens in the periodic system, the preparation of its compounds is so different from that of the other halogen compounds that it is fitting to consider the fluorine compounds in a separate section.

Inorganic fluorine compounds are prepared chiefly by the following methods:

1. Treatment of the oxides, hydroxides or carbonates with aqueous hydrogen fluoride. Most binary fluorides that do not undergo hydrolysis may be prepared in this way (e.g., alkali fluorides, alkali hydrogen fluorides, alkaline earth fluorides,  $AlF_3$ ,  $SbF_3$ ,  $ZnF_2$ ,  $PbF_2$ , HgF, AgF).

2. Treatment of the appropriate anhydrous chlorides with anhydrous hydrogen fluoride (e.g.,  $TiF_4$ ,  $ZrF_4$ ,  $NbF_5$ ,  $TaF_5$ ,  $VF_4$ ,  $SnF_4$ ,  $SbF_5$ ,  $POF_3$ ,  $SOF_2$ ). This reaction is capable of much wider application than that given in the literature until now. This method has become increasingly important since anhydrous hydrogen fluoride became commercially available.

3. Treatment of elements, oxides or halides with elemental fluorine. This method is used chiefly for the preparation of those binary fluorides in which elements reach their highest valence (e.g., IF<sub>7</sub>, ReF<sub>6</sub>, UF<sub>6</sub>, SF<sub>6</sub>, BiF<sub>5</sub>, CF<sub>4</sub>, CoF<sub>3</sub>, AgF<sub>2</sub>). The halogen fluorides may often be used instead of elemental fluorine. However, these fluorides can in turn be prepared only from elemental fluorine. There is a drawback in the use of halogen fluorides instead of elemental fluorine in that it is often difficult to separate the free halogen evolved in the reaction from the reaction product. On the other hand, most halogen fluorides are easier to handle in the laboratory (storage, measuring out) than elemental fluorine. Halogen fluorides are definitely to be preferred when, in addition to fluorine, one intends to add a second halogen to an unsaturated substance (e.g., COCIF, COBrF, COIF, SOBrF). In general, halogen fluorides may be either more active  $(ClF_3)$  or less active  $(IF_5)$  fluorinating agents than the element itself.

4. Special reactions. These are so different from each other and occur so sporadically that they cannot be classified in any systematic way. For example, NF<sub>3</sub> may be prepared by electrolysis of molten ammonium hydrogen fluoride; OF<sub>2</sub> is produced by attack of F<sub>2</sub> on 2% sodium hydroxide solution; and  $O_2F_2$  is formed from F<sub>2</sub> +  $O_2$  in a glow discharge tube cooled with liquid nitrogen.

Some fluorine compounds may be prepared by any one of several methods (e.g., NOF from  $NO+F_2$  or from  $NOBF_4 + NaF$ ), so that the choice of a method of preparation may be based on the availability of starting materials or apparatus.

The preparation of organic fluorine compounds also requires methods different from those used for the other organic halogen compounds. Because of the high heat of formation of  $CF_4$  (231 kcal.) and HF (64 kcal.), the treatment of organic substances with fluorine results mainly in the formation of  $CF_4$  and HF, in addition to charred and tarry substances, while no appreciable quantities of normal substitution products are obtained. This method, which is used in industry for the production of perfluorocarbons, is not suitable for work on a laboratory scale. Organic fluorine compounds are prepared in reasonable yields chiefly by the following procedures.

1. Addition of HF to olefins (e.g., ethyl fluoride).

2. Treatment of a chlorine compound with anhydrous hydrogen fluoride (e.g., benzotrifluoride). This method is limited to compounds with three fluorine atoms on the carbon atom. Acid fluorides can also be produced from acid chlorides, using anhydrous hydrogen fluoride.

3. Treatment of chlorine compounds with antimony(III) fluoride (F. Swarts). This method is especially suited for compounds with less than three F atoms on the carbon atom (e.g., 2,2-difluoropropane).

4. Treatment of chlorine compounds with anhydrous hydrogen fluoride in the presence of antimony catalysts. This method is intermediate between the two mentioned above and is very broadly applicable (e.g., dichlorodifluoromethane).

5. Treatment of a halogen compound with a metal fluoride, such as AgF, HgF, HgF<sub>2</sub> (e.g., fluoroform).

6. Diazotization with nitrite in hydrofluoric acid medium. This method is used for the preparation of aromatic fluorine compounds (e.g., fluorobenzene).

7. Thermal decomposition of diazonium borofluorides (G. Balz and G. Schiemann). This procedure is also applicable to aromatic fluorine compounds, particularly on the laboratory scale (e.g., p-fluorotoluene).

The laboratory equipment of the fluorine chemist is unusual in that ordinary chemical glassware cannot be used in most cases. Apparatus made of nickel, iron, copper, lead, silver, platinum, fluorspar or sintered alumina is used instead. Where transparency is indispensable, quartz equipment is utilized. To be able to produce fluorides at any time and avoid lengthy preparations, it is advisable to have on hand the following commonly used pieces of equipment:

several nickel, fluorspar or sintered alumina boats;

a nickel or Monel reaction tube (30 cm. long, 2.5 cm.

diameter) with ground joints at both ends;

an iron reaction tube;

several cylindrical iron vessels;

three to five quartz traps with attached ground joint seals; three Pyrex glass traps with attached ground joint seals; several quartz drying tubes;

several quartz U tubes;

two iron condensers;

an iron trap;

several steel cylinders, 0.5- to 5-liter capacity (for storage of gaseous fluorides).

For work with low-boiling fluorides it is useful to have on hand a part-glass, part-quartz vacuum system provided with a spiral quartz manometer.

Connection of the various parts of a quartz apparatus is best accomplished with normal, ungreased ground joints, which are made airtight by an exterior layer of cement (picein). Metal-to-quartz ground joint connections may also be made airtight with picein, but to ensure better adhesion of the cement the metal surface should be very hot. Stopcocks should be greased only lightly, preferably with viscous fluorocarbons; in difficult cases they may be replaced by copper diaphragm valves. Metal pieces may be connected with one another by means of threaded joints, flanges or ground joints. Lead rings with asbestos inserts or soft iron or copper washers may all be used as gaskets where flange connections are used. Needle valves with a steel stem, brass seat and lead-asbestos packing have proven suitable for use with steel cylinders and autoclaves.

Fluorspar apparatus is made by the following method: freshly precipitated  $CaF_2$  is mixed with water to a thick paste. To obtain plasticity, hydrochloric acid is added until the acidity of the paste is about 0.02N. It is then poured into plaster molds. Two-part molds are used for boats, three-part molds for tubing. After removal from the molds, the pieces are scraped smooth with a spatula if necessary and then air dried for a few days. Since the strength is low, all subsequent handling must be very careful. If the fluorspar apparatus cannot be fired together with ordinary ceramic ware in a tunnel kiln at about 1250°C, then it should be fired in a Globar furnace. The pieces, embedded in  $ZrO_2$ , are placed in a porcelain tube and the oven is slowly heated to 1250°C. During the firing, dry nitrogen is passed through the procelain tube to remove water vapor and carbon dioxide. Fluorspar equipment prepared in this way is nonporous and smooth. It can be worked on a wet emery wheel. In general, however, it is brittle and must be handled with great care. As a result of drying and firing the pieces shrink by about 1/3 and this should be taken into account in the design. Boats can always be made without difficulty, whereas tubes (say, of 15-cm. length, 1.5-cm. I.D., 2-mm. wall thickness) sometimes undergo deformation if the furnace temperature is somewhat too high. [O. Ruff and A. Riebeth, Z. anorg. allg. Chem. <u>173</u>, 373 (1928); O. Ruff and J. Fischer, Z. anorg. allg. Chem. <u>179</u>, 166 (1929); O. Ruff and W. Kwasnik, Z. anorg. allg. Chem. <u>209</u>, 113 (1932).]

When quartz or glass equipment is used, one must always bear in mind that the reaction products may be contaminated with flurosilicates or  $H_2SiF_S$ . Under these conditions, gaseous fluorides often contain SiF<sub>4</sub>.

Gaseous fluorides condensed by means of Dry Ice or liquid oxygen\* dissolve air in appreciable quantities, as do almost all low-boiling substances. Care must therefore be taken to remove the dissolved air by repeated distillation of the product under vacuum. The air dissolved in these low-boiling compounds may lower their melting point by as much as  $20^{\circ}$ .

General equipment of a fluorine laboratory should include large, high-suction hoods, rubber gloves, protective goggles, a gas mask, and an  $H_2-O_2$  torch for work with quartz apparatus. Dry Ice and liquid oxygen are practically indispensable for the preparation of low-boiling fluorides. One should have a bottle of 10% ammonium carbonate solution ready in case of accident. Skin injuries caused by HF or fluorides should be bathed immediately with this solution or treated with compresses containing this solution. This treatment should be administered even before medical assistance and should be continued for half an hour.

# Chlorine Monofluoride

 $\mathbf{ClF}$ 

 $\begin{array}{c} Cl_2 + F_2 = 2 ClF \\ 70.92 & 38 & 108.92 \end{array}$ 

A vertical nickel or Monel cylinder encased in a furnace (Fig. 107) serves as reaction vessel. A metered stream of chlorine gas is introduced through a nozzle-type tube. Fluorine gas is fed

<sup>\*</sup>Caution: liquid oxygen is dangerous in contact with oxidizable substances.

through a side tube. The lower part of the cylinder serves as a separator for the solid fluorides  $(NiF_2, FeF_3)$  formed by corrosion of the container walls, so that plugging is avoided. The product gases are passed through a horizontal, tapwater-cooled iron condenser and through an iron trap immersed in Dry Ice (without acetone!), where chlorine and  $ClF_3$  are condensed. The gases are then discharged into a second iron trap, immersed in liquid nitrogen. There the ClF is liquefied, while the excess fluorine escapes into the hood.



Fig. 108. Distillation of chlorine monofluoride.

The oven is heated to  $400^{\circ}$ C and  $F_{z}$  is permitted to flow through the system until it is detected at the outlet of the apparatus through

ignition of an oil-soaked rag. The chlorine stream is then turned on. For a fluorine cell current of 70 amp., the rate should be 9 liters of CL/hr.

The chlorine monofluoride, still containing considerable amounts of ClF<sub>3</sub> and Cl<sub>2</sub> in solution, is collected in the trap cooled with liquid nitrogen.

After the fluorination is ended, the CIF is distilled into a steel cylinder. The arrangement of the distillation apparatus is shown in Fig. 108. The Dewar flask in which the feed vessel is placed is lowered away from the feed vessel as far as necessary to start the product boiling. The small forerun, consisting chiefly of  $F_2$ , is removed by a water aspirator. Then the valve of the steel cylinder is opened and ClF is condensed at a rate such that the pressure in the system is maintained at about 1 atm. The distillation is ended as soon as sizable quantities of condensate (Cl.,  $ClF_3$ ) accumulate in the quartz trap. The residue is also drawn off with the water aspirator. Yield 90% maximum, based on chlorine.

PROPERTIES:

Formula weight 54.46. Colorless gas, pale yellow liquid, white solid. Very reactive, destroys glass immediately and quartz in the presence of traces of moisture. Reacts vigorously with organic substances, usually with ignition. Very vigorous reaction with water. Attacks the bronchi very strongly. M.p. -155.6°C, b.p. -100°C, t<sub>Cr</sub> -14°C, d. (liq.) (-108°C)1.67.

REFERENCES:

- O. Ruff, E. Ascher and F. Laas. Z. anorg. allg. Chem. 176, 256 (1928).
- W. Kwasnik (unpublished).

# Chlorine Trifluoride

ClF<sub>3</sub>

 $Cl_2 + 3F_2 = 2ClF_3$ 70.92 114.0 184.92

Chlorine trifluoride is prepared using the same apparatus and procedure as for ClF (Fig. 107, with the following exceptions, 1) The second liquid-nitrogen-cooled iron trap may be omitted (or left in the system without coolant). 2) The furnace is heated to 280°C instead of 400°C. 3) The chlorine flow rate is lower. For a 70-amp. current in the fluorine cell, a chlorine flow rate of
6.2 liters/hr. is advisable. If the ratio of fluorine to chlorine is too low, ClF will be the main product and the  $ClF_3$  formed will be strongly contaminated with  $Cl_2$ .

After completion of the fluorination, the liquid  $ClF_3$  is poured from the Dry Ice-cooled trap into a steel cylinder again cooled with Dry Ice (without acetone!) (use a good hood, protective goggles and rubber gloves) and a threaded shut-off valve is immediately screwed on. When the steel cylinder has warmed up to room temperature, an iron manometer and an additional valve are screwed on and the contents of the cylinder are allowed to escape until a gauge pressure of 1.2 atm. is reached. The ClF,  $Cl_2$  and  $F_2$ are thereby removed.

The yield is 60-80%, depending on the  $Cl_2: F_2$  ratio; the remainder is always CIF.

Values used for handling  $ClF_3$  must be free of all grease. Lead asbestos is suitable as value packing. Washers must be of copper. If liquid  $ClF_3$  is inadvertently spilled, Dry Ice should be sprinkled over it; it absorbs the  $ClF_3$  and dilutes it so as to render it relatively harmless.

PROPERTIES:

Formula weight 92.46. Colorless gas, suffocating odor; strongly attacks the bronchi. Extremely reactive, particularly as a liquid. Immediately destroys glass and, in presence of traces of moisture, quartz. Organic substances usually react with ignition. The reaction with water is explosive.

M.p. -83°C, b.p. 11.3°C, d. (liq.) (-78°C) 2.026.

**REFERENCES:** 

O. Ruff and H. Krug. Z. anorg. allg. Chem. 190, 270 (1930).

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 168.

### **Bromine Trifluoride**

#### BrF<sub>3</sub>

Bromine trifluoride is prepared by fluorination of bromine at  $+80^{\circ}$ C. At this temperature appreciable quantities of BrF<sub>5</sub> are formed, which then react with the excess bromine according to the equation

$$3 \operatorname{BrF}_5 + \operatorname{Br}_2 = 5 \operatorname{BrF}_3$$

thereby affording rapid and quantitative conversion of the bromine.

An inclined iron condenser (Fig. 109) serves as the reaction chamber. An iron tube, to which a dropping funnel is attached, opens into the condenser. The lower end of the condenser is joined to a wye (Y) adapter, preferably made of Monel. The adapter supports an iron reflux condenser at the top and its lower end is provided with a ground joint, to which the receiver (quartz) is connected.

Fluorine is passed through the apparatus; water at  $+80^{\circ}$ C is fed to the inclined condenser and cooling brine at  $-18^{\circ}$ C to the reflux condenser. Bromine is then added dropwise from the dropping funnel at such a rate that almost colorless BrF<sub>3</sub> is collected in the receiving flask. The reflux condenser prevents BrF<sub>3</sub> or BrF<sub>5</sub> from escaping with the fluorine stream.

At the end of the fluorination the receiver is heated for a short time to  $100^{\circ}$ C in order to remove any  $\text{BrF}_5$  which might be dissolved. After cooling, the  $\text{BrF}_3$  is poured into an iron vessel. The yield is quantitative, based on bromine, and 90%, based on fluorine. The purity is 98% (the remainder being  $\text{BrF}_5$ ).



Fig. 109. Preparation of bromine trifluoride.

PROPERTIES:

Formula weight 136.91. Colorless liquid. Very reactive; fumes in air and strongly attacks the skin.

M.p. +8.8°C, b.p.+127°C; d. (liq.) 2.84. Crystal form: long prisms.

**REFERENCE:** 

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 168.

### **Bromine Pentafluoride**

BrF₅

 $2 Br + 5 F_2 = 2 Br F_5$ 159.82 190 349.82

An iron reaction vessel is placed in a crucible furnace held at  $200^{\circ}$ C (Fig. 110). In addition to the inlet tube for  $F_2$  and the outlet tube for  $BrF_5$ , the reaction vessel is provided with a thermometer well and a central wide T tube adapter, which holds a dropping funnel and permits the nitrogen to flow in from the side. (The nitrogen flow serves merely as a purge, to keep the fluorine penetrating the dropping funnel.) The gaseous reaction products are condensed in an iron condenser. The liquefied  $BrF_5$  collects in an iron trap, which is cooled with an ice-salt mixture.

As soon as the apparatus is filled with fluorine, dropwise input of  $Br_2$  into the reaction vessel is started. About one drop per second of bromine is introduced at a fluorine cell current of 150 amp. Care must be taken to maintain a steady excess of  $F_2$ .



Fig. 110. Preparation of bromine pentafluoride.

The crude product consists of 95%  $BrF_5$  and 5%  $BrF_3$ . After the fluorination is terminated, the product is distilled from an iron apparatus in a stream of fluorine. Condenser and receiver are cooled to about  $-18^{\circ}C$  with an ice-salt mixture. The yield is 87%, based on bromine. Bromine pentafluoride is stored in iron or, preferably, in Monel vessels.

#### PROPERTIES:

Formula weight 174.91. Colorless liquid; fumes strongly in air. Completely stable up to 460°C. M.p.  $-61.3^{\circ}$ C, b.p.  $+40.5^{\circ}$ C; d. (liq.) (0°C) 2.57.

Very reactive; reacts with nearly all elements with ignition. Vigorous, nearly explosive reaction with water. At room temperature, dry glass is attacked slowly, quartz glass practically not at all. Mercury becomes coated with a brown film.

**REFERENCES:** 

O. Ruff and W. Menzel. Z. anorg. allg. Chem. <u>202</u>, 49 (1931). W. Kwasnik (unpublished).

#### **Iodine Pentafluoride**

 $IF_5$ 

 $I_2 + 5 F_2 = 2 I F_5$ 253.84 190 443.84

An iron drum (see Fig. 111) provided with a cooling jacket serves as reaction vessel. It connects to an inclined iron condenser, followed by two quartz traps.



Fig. 111. Preparation of iodine pentafluoride.

The reaction vessel is filled with iodine, and  $F_2$  (preferably HFfree) is passed through. The coolant removes the heat of reaction and prevents conversion of the IF<sub>5</sub> remaining in the vessel to IF<sub>7</sub>. As soon as  $F_2$  is detected at the outlet of the system, the reaction should be terminated. Cooling of the reaction vessel is stopped, heat is applied with a gas burner so that the jacket serves as a water bath, and IF<sub>5</sub> is distilled off in a stream of fluorine. At a fluorine flow rate corresponding to a cell current of 80 amp., 10 hours are necessary for the conversion of 250 g. of iodine (including distillation). The yield is 90%, based on iodine. Iodine pentafluoride is stored in iron flasks. It is well suited for the fluorination of organic compounds.

PROPERTIES:

Formula weight 221.92. Colorless liquid; fumes in air, reacts very vigorously with water.

M.p. +9.6°C, b.p. +98°C; d. (liq.) (15°C) 3.231, d. (solid) (0°C) 3.75.

**REFERENCES:** 

F. Moissan. Bull. Soc. chim. France [3] 29, 6 (1930). W. Kwasnik (unpublished).

### lodine Heptafluoride

IF7

 $I_2 + 7 F_2 = 2 I F_{\tau}$ 253.84 266 519.84

An iron cylinder provided with a cooling jacket (see Fig. 112) serves as the reaction vessel. The inlet opening is provided with a strainer to hold the iodine. The outlet opening has an appr. 30-cm.-long adapter to which an iron condenser is screwed on. An iron



Fig. 112. Preparation of iodine heptafluoride.

tube leads from the latter to the U-shaped condensation traps. These are made of quartz and are closed at the top with loosely fitting quartz ground joints. The first U tube is used to trap the  $IF_{\gamma}$ , and

the second to exclude atmospheric moisture. Both tubes are cooled with liquid nitrogen.

The screen-type strainer in the reaction vessel is filled with iodine, and fluorine is passed through the system. The  $F_2$  must be HF-free. This can best be achieved either by passing it through a DryIce-cooled iron coil or over freshly dehydrated KF. In the first stage of the reaction the iodine burns in the reaction vessel to IF<sub>5</sub>. While the reaction is in progress, a high water flow rate must be maintained in the cooling jacket. As soon as fluorine appears at the outlet of the system, the flow of water through the jacket is shut off, and the jacket is heated from the outside with a gas flame so that it acts as a water bath. The adapter above the reaction vessel is now electrically heated to about 300°C. The fluorine stream converts the IF<sub>5</sub> into IF<sub>7</sub>, which escapes through the vertical condenser into the quartz trap. This condenser must be very efficient so as to retain the unconverted IF<sub>5</sub>. At this stage of the reaction, fluorine must be present in excess.

The solid product collecting in the arms of the U tubes is melted down from time to time or pushed down with an iron wire. To do this, the ground stoppers must be removed for a few seconds.

The IF<sub>7</sub> is purified by distillation at atmospheric pressure and  $+40^{\circ}$ C and is collected in quartz traps at  $-196^{\circ}$ C. The small residue is IF<sub>5</sub>, which is returned to the reactor. The IF<sub>7</sub> is then distilled into a steel cylinder. When the filled cylinder reaches room temperature, the valve is carefully opened, and the gas (chiefly SiF<sub>4</sub>) is vented until a wad of alcohol-soaked cotton wool is ignited by the escaping IF<sub>7</sub>. The yield is 83%, based on iodine.

Ground joint connections easily freeze on contact with iodine fluorides. They should therefore be moved a little from time to time. The joints should not be greased nor should they be sealed with picein; they should only be loosely fitting.

#### PROPERTIES:

Formula weight 259.84. Colorless. In the solid state it appears as a loose powder, sometimes in the form of small crystals.

M.p.  $5.5^{\circ}$ C, subl. p.  $4.5^{\circ}$ C; d. (liq.) ( $6^{\circ}$ C) 2.8.

Very reactive. Similar in properties to  $ClF_3$ , but considerably less reactive. Water dissolves gaseous IF, without detonation. Sodium hydroxide solution absorbs it with evolution of a large amount of heat. Sulfuric acid foams when IF<sub>7</sub> is bubbled through it. Attacks glass and quartz. Musty, acidic odor.

#### REFERENCES:

O. Ruff and R. Keim. Z. anorg. allg. Chem. <u>193</u>, 176 (1930). W. Kwasnik (not yet published).

### Dioxygen Difluoride

 $O_2F_2$ 

 $O_2 + F_2 = O_2 F_2$ 32.0 38.0 70.0

The gases are electrically excited in a discharge tube cooled with liquid nitrogen, and the unstable  $O_2F_2$  thus formed is frozen out.

The fluorine is stored in a quartz trap cooled with liquid nitrogen and is aspirated into the apparatus through a copper diaphragm valve. If available, a steel cylinder with fluorine may be connected directly. The oxygen is also taken from a steel cylinder, which is connected to the system through an iron or copper capillary. The reaction vessel is a glass flask (see Fig. 113) immersed in a Dewar flask filled with liquid nitrogen; provisions are made for generation of a brush discharge. Copper wires, tightly cemented with picein into narrow glass tubes projecting about 10 cm. outside the apparatus, serve as electrodes. The gas discharge is generated by a large induction coil with either a Wehnelt interrupter or an a.c. transformer, whose secondary supplies 0.05 amp. at about 5000 v. A wide-arm quartz U tube is attached to the discharge tube. After completion of the reaction it serves as distillation receiver and storage container. A quartz trap, cooled to  $-196^{\circ}C$ 



Fig. 113. Preparation of dioxygen difluoride.

to prevent access of atmospheric moisture, is attached to the U tube. A metal aspirator is best for the generation of a vacuum. A manometer for the measurement of the vacuum is not necessary, since the shape of the gas discharge gives a good indication of the vacuum attained. The best operating pressure range is 10-20 mm. In this range the discharge takes on a brushlike shape.

As soon as the Dewar flasks have been placed under the corresponding parts of the apparatus and the aspirator has been started, the spark coil is switched on. The diaphragm valve is then carefully opened to allow fluorine to pass into the system. Only then is oxygen allowed to flow in, but at a lower than stoichiometric rate, for otherwise solid ozone (violet to blue) may form in the receiving flask. The  $O_2F_2$  separates out on the walls of the discharge tube as a red-brown solid. From time to time the electric discharge is interrupted for a few minutes and the Dewar flask is lowered. This allows the  $O_2F_2$  to melt and flow down into the lower tubular extension of the reaction vessel. If solid ozone is present, the melting should be done very carefully since explosions may sometimes occur.

As soon as enough  $O_2F_2$  has accumulated in the bottom tip of the discharge tube, the spark coil is disconnected and the oxygen and fluorine streams are turned off. The U-shaped storage vessel is cooled with liquid nitrogen and the Dewar flask is removed from beneath the discharge tube. The  $O_2F_2$  distills over (at about 15 mm.) with partial decomposition. To minimize decomposition, care should be taken not to let the  $O_2F_2$  temperature exceed  $-60^{\circ}C$  for more than a short time during the distillation. The  $O_2F_2$  may be redistilled several times in this manner to ensure purity. The first cuts are mainly ozone and SiF<sub>4</sub>. Since the distillation always results in accumulation of  $O_2$  and  $F_2$ , the aspirator should be kept continuously in operation. For the same reason, difficulties are encountered in the distillation of  $O_2F_2$  into ampoules.

After distillation is completed, the U-shaped storage vessel is melt-sealed at the bottom sections of the arms. Dioxygen difluoride can be stored only in liquid nitrogen. In emergencies, Dry Ice may be used.

PROPERTIES:

Brown gas, cherry-red liquid, and orange solid.

M.p.  $-163.5^{\circ}$ C, b.p.  $-57^{\circ}$ C; d(liq.)( $-57^{\circ}$ C) 1.45, d(solid)( $-165^{\circ}$ C) 1.912.

**REFERENCES:** 

O. Ruff and W. Menzel. Z. anorg. allg. Chem. <u>211</u>, 204 (1933).
 O. Ruff and W. Menzel. Z. anorg. allg. Chem. <u>217</u>, 85 (1934).

### Oxygen Difluoride

OF2

 $\begin{array}{rrrr} 2\,F_2 + 2\,NaOH = OF_2 + 2\,NaF + H_2O\\ 76 & 80.0 & 54 & 83.99 & 18.01 \end{array}$ 

Fluorine gas is bubbled at a rate of 1 to 3 liters/hour from a platinum tube of appr. 2 mm. I.D. into a 2% solution of NaOH

contained in a glass reaction vessel. The NaOH solution flows through the reaction vessel at a rate of 1 liter/hour from an elevated storage container. The platinum tube dips about 2 cm. into the sodium hydroxide solution (Fig. 114).

The output gas mixture flows through wash bottle filled with water, which absorbs the unreacted fluorine. The  $OF_2$  is then condensed in two glass traps immersed in liquid nitrogen.

After completion of the reaction, the crude product, condensed at  $-196^{\circ}$ C in the traps, is evacuated with a water aspirator to a pressure of 20 mm., which removes the major portion of the oxygen dissolved in the OF<sub>2</sub>. To prevent simultaneous escape of the OF<sub>2</sub> into the atmosphere, a wash bottle containing KI solution is inserted before the aspirator. The OF<sub>2</sub> is then fractionated, with oxygen coming over first. The first distillation gives a 98.5% pure product. The yield is 45%, based on fluorine.

Oxygen difluoride is stored in glass flasks or steel cylinders.



Fig. 114. Preparation of oxygen difluoride.

#### PROPERTIES:

Colorless gas; yellow, brownish-tinged liquid.

M.p.  $-223.8^{\circ}$ C, b.p.  $-144.8^{\circ}$ C; t<sub>Cr</sub>  $-58.0^{\circ}$ C, p<sub>Cr</sub> 48.0 abs. atm, c<sub>Cr</sub> 97.6 ml./mole; d (liq.) ( $-223.8^{\circ}$ C) 1.90, ( $-145.3^{\circ}$ C) 1.521;  $\Delta$ H (formation)-11 kcal. Solubility in water at 0°C: 6.8 ml. of gaseous OF<sub>2</sub>/100 ml.

Characteristic odor. Inhalation causes severe breathing difficulties, which often do not begin until several hours after inhalation and persist for hours. Does not attack glass. Reaction with water is hardly noticeable. Attacks mercury.

Stable to light, heat and electrical ignition. Remarkably unreactive compared to  $Cl_2O$ . Like all low-boiling fluorides, the liquid dissolves appreciable quantities of air.

REFERENCES:

- P. Lebeau and A. Damiens. Compt. Rend. Hebd. Séances Acad. Sci. 188, 1253 (1938).
- O. Ruff and W. Menzel. Z. anorg. allg. Chem. 190, 257 (1930).

### Chlorine Dioxide Fluoride

#### ClO<sub>2</sub>F

 $\begin{array}{rl} 2\,\text{ClO}_2 + \,F_2 = 2\,\text{ClO}_2 F \\ 134.92 & 38 & 172.92 \end{array}$ 

The apparatus, which must be made entirely of quartz, is set up as in Fig. 115. Fluorine flows at a rate of 500 ml./hr. into the first trap, in which a few milliliters of liquid  $ClO_2$  at -50 to  $-55^{\circ}C$ have been placed. The inlet tube dips a few millimeters into the liquid  $ClO_2$ . The reaction progresses smoothly and steadily; most of the  $ClO_2F$  formed in the reactor remains there and only a small portion reaches the second trap. When the color of the liquid in the first trap becomes very faint, the reactor is allowed to warm and the  $ClO_2F$  is distilled into the second trap in a stream of fluorine, with gradually rising temperature. It collects as a pure, colorless substance requiring no further purification.



Fig. 115. Preparation of chlorine dioxide fluoride.

According to M. Schmeisser, the procedure may be advantageously altered in the following way:  $ClO_2$  is dissolved in  $CCl_3F$  at  $-78^{\circ}C$  and fluorinated. The  $ClO_2F$  formed separates as a denser liquid phase when saturation is reached. The mixture is then cooled to  $-110^{\circ}C$  and the less dense liquid phase is rapidly removed by vacuum through a capillary. The reaction may be performed in Pyrex glass apparatus. If absolutely pure  $ClO_2F$  is required, working in quartz apparatus without excess

ClO<sub>2</sub> becomes a necessity and entails repeated rectification of the product.

SYNONYM:

Chloryl fluoride.

PROPERTIES:

Formula weight 86.46. Colorless, very sensitive to moisture, immediately forms a fog in moistair. Thermally much more stable than  $ClO_2$ .

M.p.  $-115^{\circ}$ C, b.p.  $-6^{\circ}$ C.

REFERENCES:

- H. Schmitz and H. J. Schumacher. Z. anorg. allg. Chem. <u>249</u>, 242 (1942).
- J. E. Sicre and H. J. Schumacher. Z. anorg. allg. Chem. <u>286</u>, 232 (1956).
- M. Schmeisser and F. L. Ebenhöch. Angew. Chem. 66, 230 (1954).

### Chlorine Trioxide Fluoride

#### ClO<sub>8</sub>F

 $\begin{array}{rl} KClO_4 \,+\, HSO_3F \,=\, ClO_3F \,+\, KHSO_4 \\ 138.56 & 100.07 & 102.46 & 136.17 \end{array}$ 

Ten grams of KClO<sub>4</sub> are dissolved in 100 g. of HSO<sub>3</sub>F in a roundbottom glass flask provided with stirrer and reflux condenser. The reaction starts at 50°C and is complete at 85°C. The reaction gases are allowed to pass over a 10% sodium hydroxide solution containing 5% Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> and are then bubbled through a similar solution. The gas is dried with solid KOH and then condensed in a trap cooled with liquid nitrogen. During the reaction a stream of dry nitrogen is bubbled through the reaction mixture. The product contains 1% air and 0.4% CO<sub>2</sub>. This procedure is also suited to the production of ClO<sub>3</sub>F in kilogram quantities.

SYNONYM:

Chloryl oxyfluoride.

PROPERTIES:

Colorless gas, with a characteristic odor reminiscent of  $OF_2$ . M.p. -152.2°C, b.p. -48.1°C. Fairly stable thermally. May be heated in glass nearly to the softening point. Somewhat soluble in water. Reacts quite slowly with dilute aqueous alkali.

REFERENCES:

G. Barth-Wehrenalp. J. Inorg. Nucl. Chem. 2, 266 (1956).

# Chlorine Tetroxide Fluoride ClO4F

 $\begin{array}{l} \text{HClO}_4 + \text{F}_2 = \text{ClO}_4\text{F} + \text{HF} \\ \text{100.46} \quad 38.0 \quad 118.46 \quad 20.0 \end{array}$ 

The apparatus (see Fig. 116) is best made of quartz. It consists of a cylindrical reaction tube appr. 50 cm. high, filled with quartz Raschig rings. The tube is surrounded by a cooling jacket fed with flowing water. The fluorine is introduced through a quartz tube reaching nearly to the bottom of the vessel, and 70%  $\rm HClO_4$  is added from a dropping funnel. The liquid is drained through a siphon, while the gaseous reaction products are drawn off by suction from the top of the reaction tube into a quartz trap immersed in liquid nitrogen. The system ends with a drying tube filled with anhydrous KF.



Fig. 116. Preparation of chlorine tetroxide fluoride.

The  $HClO_4$  input is one drop per second, until the liquid begins to flow through the siphon. Fluorine is then introduced at a rate of 2.5 liters/hour. In addition to  $OF_2$ ,  $Cl_2$  and  $SiF_4$ , solid  $ClO_4F$  collects on the walls of the condensing trap. Because of the high explosion hazard,  $ClO_4F$  should never be made in quantities larger than 4 g. The product is purified by fractional vacuum distillation. It is stored in quartz ampoules cooled with liquid nitrogen. Explosions may easily occur when  $ClO_4F$  is melted and solidified. The yield is about 60%.

The highest yields (over 90%) are obtained with a platinum reaction vessel. Glass may be used as the apparatus material if other material is unavailable, but the product is then quite impure and obtained in low yield. Carbon cannot be used as the construction material, since it catalytically decomposes  $ClO_4F$ .

PROPERTIES:

Colorless gas, very explosive; often explodes during melting or condensation. Gaseous  $ClO_4F$  explodes in a manner similar to  $NO_3F$ , through mere contact with dust, grease, rubber or 2N KI solution. In an open beaker, gaseous  $ClO_4F$  explodes upon contact with a flame or spark. It has a strongly acrid odor, irritates the throat and lungs, and causes persistent respiratory trouble.

M.p. -167.2°C, b.p. -15.9°C.

**REFERENCES:** 

G. H. Rohrback and G. H. Cady. J. Amer. Chem. Soc. 69, 677 (1948).

### Sulfur Tetrafluoride

 $SF_4$ 

 $4 \operatorname{CoF}_3 + S = SF_4 + 4 \operatorname{CoF}_2$ 463.76 32.06 108.06 387.86

I. Sulfur (17 g.) is introduced into a quartz flask and covered with a layer of dry calcium fluoride powder (40 g., 0.025 to 0.05 mm. diameter grains), and 270 g. of CoF<sub>3</sub> is placed on top. The flask is connected with a short piece of tubing to a trap immersed in liquid nitrogen. This in turn is connected to a vacuum system. The apparatus is now evacuated and the quartz flask is shaken so that the two raw materials and the calcium fluoride are mixed together. Reaction begins during the mixing, with evolution of gas. The temperature is gradually raised to  $130^{\circ}$ C by means of an oil bath and maintained for two hours. A colorless product condenses in the trap during the run. This is subsequently fractionated, using a quartz spiral manometer to permit control at zero mm. gage. Sulfur hexafluoride (up to 6% of the total quantity) comes over in the first cut. The main cut is  $SF_4$ , which contains small quantities of lower sulfur fluorides (SF<sub>2</sub>,  $\dot{S}_2F_2$ ). In order to obtain absolutely pure SF<sub>4</sub>, it is shaken with mercury in a platinum (not quartz) flask; this removes the lower sulfur fluorides. The degree of purity is ascertained by determining the molecular weight by the vapor density method. The boiling point is not decisive for estimating the degree of purity of  $SF_4$ .

II. SF<sub>4</sub> may be obtained in smaller yields (40%), according to F. Brown and P. L. Robinson, by careful fluorination of sulfur with fluorine at  $-70^{\circ}$ C (F<sub>2</sub>: N<sub>2</sub> = 1 : 3).

Sulfur tetrafluoride is stored in sealed quartz ampoules.

**PROPERTIES:** 

Colorless gas, thermally stable up to  $-600^{\circ}$ C; reacts vigorously with water. Decomposed exothermally by concentrated H<sub>2</sub>SO<sub>4</sub>.. Attacks glass but not quartz or mercury.

M.p. -121.0°C, b.p. -40.4°C; d(liq.) (-78°C) 1.95, (solid) (-183°C) 2.349. Readily soluble in benzene.

REFERENCES:

- I. W. Luchsinger. Thesis, Techn. Hochschule, Breslau, 1936, p. 23.
- II. F. Brown and P. L. Robinson. J. Chem. Soc. (London) <u>1955</u>, 3147.

#### Sulfur Hexafluoride

SF6

I.

 $S + 3 F_2 = SF_6$ 32.06 114.0 146.06

The reactor is a nickel tube 300 mm, long and 25 mm. I.D. (Fig. 117) containing a nickel boat filled with sulfur. The ground joints of the reaction tube and of the quartz trap are best left ungreased and uncemented but only tightly compressed. The iron drying tube containing freshly dehydrated KF is for exclusion of moisture. The quartz trap is cooled with liquid nitrogen.

This apparatus is suitable for most fluorinations in which a solid raw material forms a gaseous fluoride (SeF<sub>6</sub>, TeF<sub>6</sub>, AsF<sub>5</sub>, CF<sub>4</sub>, GeF<sub>4</sub>, MoF<sub>6</sub>, WF<sub>6</sub>). In the special case of SF<sub>6</sub> the apparatus may be made entirely of glass.

The sulfur burns with a bluish flame in the fluorine stream. The product collects in the condensation trap and is then passed



Fig. 117. Preparation of sulfur hexafluoride.

through fritted wash bottles containing hot 10% KOH (not NaOH) in order to remove impurities (HF, SF<sub>2</sub>, SF<sub>4</sub>, SOF<sub>2</sub>, S<sub>2</sub>F<sub>10</sub>). Finally, the gas is dried in a  $P_2O_5$  tube and is passed at room temperature over activated charcoal to remove  $S_2F_{10}$ . The yield is 87%.

II. 
$$SO_2 + 3F_2 = SF_6 + O_2$$
  
118.97 40 86.06 72.92

Sulfur dioxide is burned with an excess of fluorine to  $SF_g$  in the apparatus described for the preparation of  $COF_2$  (p. 206 f). The temperature should be as high as possible, preferably about  $650^{\circ}C$ . The chief impurity in the crude  $SF_g$  accumulating in the condensation trap is  $SO_2F_2$ . The  $SF_g$  is passed through several fritted wash bottles filled with water and hot 10% KOH, and then dried over  $P_2O_5$ . The yield is 70%, based on  $SO_2$ .

Sulfur hexafluoride may be stored in a gasometer over water, in a glass flask provided with a stopcock, or, under pressure, in a steel cylinder.

**PROPERTIES:** 

Colorless, odorless; thermally and chemically very stable. M.p.  $-50.8^{\circ}$ C (under pressure), subl. p.  $-63.8^{\circ}$ C, t<sub>Cr</sub>  $+45.55^{\circ}$ C, p<sub>Cr</sub> 38.33 abs. atm; d (liq.) ( $-50.8^{\circ}$ C) 1.88.

Very sparingly soluble in water, slightly soluble in alcohol.

**REFERENCES:** 

I. W. Klemm and P. Henkel. Z. anorg. allg. Chem. 207, 73 (1932).

II. German Patent Application I. 72173 IV b/12 i, May 4, 1942;
 W. Kwasnik, not yet published.

### Thionyl Fluoride

#### SOF<sub>2</sub>

 $SOCl_2 + 2 HF = SOF_2 + 2 HCl$ 118.97 40 86.06 72.92

An iron bottle (Fig. 118) with a gas inlet tube serves as the reaction vessel. A second bottle is connected to the first, to retain the unreacted HF. This is joined to a glass gas trap immersed in liquid nitrogen or a Dry Ice-acetone bath. A drying tube filled with KF is attached to exclude atmospheric ministure.

The reaction vessel is filled with 500 g. of  $SOCl_{2}$  and 50 g. of  $SbCl_{5}$  (catalyst), and anhydrous gaseous HF is introduced through



Fig. 118. Preparation of thionyl fluoride.

the inlet tube. The HF is thoroughly absorbed, causing a mixture of  $SOF_2$  and HCl to be evolved, which then is collected in the condensation trap. The reaction is so endothermic that the outside of the reaction vessel gradually becomes covered with ice. When all the  $SOCl_2$  has been consumed, more may be added without further addition of  $SbCl_z$ .

Separation of the  $SOF_2$  from the HCl may be achieved either by distillation or by rapid bubbling of the gas mixture through ice-cold water, in which HCl is completely absorbed, while the SOF<sub>2</sub> passes through with almost no decomposition. The gas is then dried over concentrated H<sub>2</sub>SO<sub>4</sub> or over P<sub>2</sub>O<sub>5</sub>.

Thionyl fluoride is stored under pressure in steel cylinders. PROPERTIES:

Colorless gas, thermally stable up to red heat. Does not corrode Fe, Ni, Co, Hg, Si, Mn, B, Mg, Al or Zn below 125°C. Does not attack glass. Suffocating odor. Hydrolyzed very slowly in ice-cold water.

M.p.  $-110.5^{\circ}$ C, b.p.  $-43.7^{\circ}$ C, t<sub>cr</sub>  $+88^{\circ}$ C; d. (liq.) ( $-100^{\circ}$ C) 1.780, d. (solid) ( $-183^{\circ}$ C) 2.095.

#### **REFERENCES:**

German Patent Application I. 53743 IV b/12 i.

J. Söll and W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 192.

H. S. Booth and F. C. Merciola. J. Amer. Chem. Soc. <u>62</u>, 640 (1940). U. Wannagat and G. Mennicken, Z. anorg. allg. Chem. <u>278</u>, 310 (1955).

# Thionyl Tetrafluoride

SOF<sub>4</sub>

$$SOF_2 + F_2 = SOF_4$$
  
86.06 38 124.06

The apparatus (Fig. 119) is equipped with a nickel T tube for gas mixing. The T tube opens into a largernickel tube, which is heated in an electric furnace. The output reaction gases pass through two quartz traps. The product  $SOF_4$  is frozen out in these traps at  $-196^{\circ}C$ . The system ends in a drying tube filled with freshly dehydrated KF to exclude atmospheric moisture.



Fig. 119. Preparation of thionyl tetrafluoride.

This arrangement may be used for most fluorinations involving participation of two gases (NO +  $F_2$ , NO<sub>2</sub> +  $F_2$ , CrO<sub>2</sub>Cl<sub>2</sub> +  $F_2$ , PF<sub>3</sub> + Cl<sub>2</sub>).

A platinum wire-screen spiral is introduced into the reaction tube, and the furnace is heated to  $150^{\circ}$ C. If no screen is available, the reaction may still be carried out, but the furnace must then be held at 300°C. The reaction tube may be made of quartz, but nickel is definitely preferred. Fluorine and SOF<sub>2</sub> streams are mixed in a 1.1:1 ratio. An efficient way to do this is to calculate the quantity of F<sub>2</sub> per hour from the current load on the fluorine generator and to meter an appropriate quantity of SOF<sub>2</sub> per hour by means of a differential manometer flowmeter (cf. Part I, p. 85 or H. Lux, Anorg. Chem. Experimentierkunst [The Art of Experimentation in Inorganic Chemistry], Leipzig 1959, p. 450). Concentrated H<sub>2</sub>SO<sub>4</sub> or, better, liquid paraffin are suitable as manometric fluids. The SOF<sub>4</sub> collects as a white solid in the condensation traps.

After completion of the reaction, the  $SOF_4$  is purified by fractional distillation in a quartz apparatus.

Thionyl tetrafluoride is stored under pressure in steel cylinders or in glass ampoules cooled with Dry Ice or liquid nitrogen.

PROPERTIES:

Colorless gas, pungent odor. Highly exothermic reaction with water with formation of  $SO_{2}F_{2}$ . Completely absorbed by NaOH solution. Pure SOF<sub>4</sub> does not attack glass. M.p. -99.6°C, b.p. -48.5°C, d(liq.) (-82°C) 1.946, d(solid)

M.p.  $-99.6^{\circ}$ C, b.p.  $-48.5^{\circ}$ C, d(liq.) ( $-82^{\circ}$ C) 1.946, d(solid) ( $-183^{\circ}$ C) 2.55.

**REFERENCES:** 

O. Ruff and H. Jonas. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 192.

German Patent Application R100449.

H. Jonas. Z. anorg. allg. Chem. 265, 273 (1951).

### Sulfuryl Fluoride

SO<sub>2</sub>F<sub>2</sub>

 $\begin{array}{rl} BaCl_2 \ + \ 2 \ HSO_3F \ = \ Ba(SO_3F)_2 \ + \ 2 \ HCl \\ 208.27 \ & 200.14 \ & 335.48 \ & 72.92 \end{array}$ 

 $\begin{array}{rl} Ba(SO_3F)_2 = SO_2F_2 + BaSO_4 \\ & 335.48 & 102.06 & 233.42 \end{array}$ 

Barium chloride is dehydrated by heating to  $200^{\circ}C$  and then pulverized. The powder (100 g.) is added little by little to 100 g. of  $HSO_3F$ , placed in an ice-cooled iron vessel (about 500-ml. capacity) with a screw lid and agas outlet tube. The reaction is very vigorous, and a stream of HCl is evolved. As soon as all the BaCl<sub>2</sub> is added, the lid is screwed on and the vessel heated at  $100^{\circ}C$  until no more HCl vapor escapes. The iron vessel is then connected to a water aspirator and heated to  $120-150^{\circ}C$  under vacuum, to remove any excess of HSO<sub>3</sub>F and the last traces of HCl.

A glass gas trap is now connected to the gas outlet tube and immersed in liquid nitrogen while the iron vessel is further heated. Decomposition of the barium fluorosulfonate begins at  $400^{\circ}$ C and becomes vigorous at  $450-500^{\circ}$ C.

The condensate accumulating in the trap is then passed through a wash bottle containing warm  $\rm KMnO_4$  solution (to remove  $\rm SO_2$ ), then through a second wash bottle with concentrated  $\rm H_2SO_4$ , then through a drying tube containing  $\rm P_2O_5$ , and finally is again condensed in a trap at --196°C. The product is now distilled, and the first and last cuts discarded. The yield is 60%, based on BaCl<sub>2</sub>.

Sulfuryl fluoride is stored in a gas holder over concentrated  $H_2SO_4$  or compressed into steel cylinders.

PROPERTIES:

Colorless, odorless gas, thermally stable up to 400°C; chemically very unreactive, not hydrolyzed by water, dissolves fairly rapidly in alkali hydroxide solution, with complete hydrolysis.

M.p. -121.4°C, b.p. -49.7°C; d. (liq.) about 1.7.

Solubility at 16.5°C in water: 4- 5 ml. gaseous  $SO_2F_2/100$  ml. in alcohol: 24- 27 ml. gaseous  $SO_2F_2/100$  ml. in toluene: 210-220 ml. gaseous  $SO_2F_2/100$  ml. in CCl<sub>4</sub>: 136-138 ml. gaseous  $SO_2F_2/100$  ml.

**REFERENCE:** 

M. Trautz and K. Ehrmann. J. prakt. Chem. (N.S.) 142, 91 (1935).

### Trisulfuryl Fluoride

S<sub>3</sub>O<sub>8</sub>F<sub>2</sub>

 $3 SO_3 + BF_3 = S_3O_8F_2 + BOF_{240.618} 67.82 262.18 45.82$ 

Liquid SO<sub>3</sub> is saturated with BF<sub>3</sub>. The liquid becomes cloudy due to formation of a precipitate which is difficult to filter. The reaction mixture is then treated with 70% sulfuric acid, while being cooled with ice, and a heavy, colorless liquid phase separates. This is washed with concentrated  $H_2SO_4$ ; it is then of reagent grade.

PROPERTIES:

Fumes in air, insoluble in concentrated H<sub>2</sub>SO<sub>4</sub>, hydrolyzes very slowly in dilute potassium hydroxide because of formation of a salt film of K<sub>2</sub>SO<sub>4</sub> and KSO<sub>3</sub>F at the contact area. B,p. 120°C (dec.):d<sup>25</sup> 1.86.

REFERENCES:

H. A. Lehmann and L. Kolditz. Z. anorg. allg. Chem. 272, 73 (1953).

### Thionyl Chloride Fluoride

#### SOCIF

 $\begin{array}{rrrr} 4\, SOCl_2 + IF_5 = 3\, SOClF + \, SOF_2 + ICl_3 + Cl_2 \\ 475.9 & 221.92 & 307.5 & 86.06 & 233.29 & 70.92 \end{array}$ 

A flask provided with a reflux condenser and a dropping funnel (all made of quartz) is filled with 60 g. of  $SOCl_2$ , and 45 g. of  $IF_5$  is slowly added dropwise from the funnel. Heat is evolved and the color darkens. The gases escaping through the reflux condenser

174

are collected in a quartz trap immersed in liquid nitrogen. At the outlet of the system there is a drying tube, filled with anhydrous KF.

The product condensed in the gas trap is greenish-yellow from the entrained  $ICl_3$ . It is distilled over antimony powder or mercury until colorless. It is then fractionated, with SiF<sub>4</sub>, HCl and SOF<sub>2</sub> coming over first and SOCl<sub>2</sub> last. The fraction collected between 10 and 18°C (760 mm.) is SOC1F. The yield is about 42%, based on SOCl<sub>2</sub>.

Thionyl chloride fluoride is best stored in glass ampoules cooled with liquid nitrogen; for short periods it may, if necessary, be kept at room temperature in glass flasks or steel cylinders.

PROPERTIES:

Colorless gas; decomposes at room temperature into  $SOCl_2 + SOF_2$ ; choking odor similar to the sulfur fluorides. Disproportionates at room temperature in contact with Cu and Hg, and with Fe above  $70^{\circ}C$ . Water and sodium hydroxide solution cause hydrolysis. Does not attack glass.

Broad melting range between -110 and  $-139^{\circ}C$  (mixture of two isomers), b.p. 12.3°C; d. (liq.) (0°C) 1.576.

REFERENCES:

O. Ruff and H. Jonas. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 192.

German Patent Application R 100449.

H. Jonas. Z. anorg. allg. Chem. 265, 273 (1951).

# Sulfuryl Chloride Fluoride

SO<sub>2</sub>ClF

I.

$3 SO_2Cl_2$	+ SbF <sub>3</sub> =	3 SO <sub>2</sub> ClF	+ SbCl <sub>3</sub>
404.91	178.76	355.5	228.13

The reaction vessel is a one-liter autoclave or steel cylinger with a screwed-on water-cooled reflux condenser. The condenser is equipped at its upper end with a spring-type manometer and a blowoff valve. The equipment must be able to withstand a pressure of 10 atm. gage.

The blowoff valve is connected to two quartz traps immersed in liquid nitrogen. At the outlet there is a drying tube with anhydrous KF to exclude atmospheric moisture.

The reaction vessel is filled with 220 ml. (365 g.) of  $SO_2Cl_2$ , 187 g. of finely divided  $SbF_3$ , and 40 ml. of  $SbCl_5$  (catalyst).

Heat is gradually applied up to  $300^{\circ}$ C, whereby a pressure of 7 atm. gage builds up. The reaction gases are allowed to escape slowly into the quartz traps by slowly opening the valve until the pressure in the reaction vessel is 6.3 atm. gage. In this way, about 80 ml. of condensate collects in the traps within two hours.

The product is then distilled, with HCl and  $SO_2$  coming over as the first fraction while the last cut consists of unconverted  $SO_2Cl_2$ . The yield is 50 ml. of pure  $SO_2ClF$ .

II.  $SO_2Cl_2 + HF = SO_2ClF + HCl_{134.97} 20 118.5 36.46$ 

Technical grade  $SO_2Cl_2$  (900 g.), 130 g. of anhydrous HF, 200 g. of SbF<sub>3</sub>, and 40 ml. of SbCl<sub>5</sub> are introduced into an autoclave equipped with a fractionating column, and the contents are heated to 250-300°C. A pressure of 40 to 50 atm. gage builds up and is maintained by slowly releasing HCl through the water-cooled column. The reaction is complete after two to four hours of heating. The  $SO_2ClF$  is distilled from the autoclave into quartz traps, as described in method I. The  $SbCl_5$  catalyst may be reused. The yield is 80 to 85%, based on  $SO_2Cl_2$ .

Sulfuryl chloride fluoride is stored in steel cylinders or glass flasks.

PROPERTIES:

Colorless gas, pungent odor similar to SO<sub>2</sub>Cl<sub>2</sub>, does not fume in air, reacts rapidly with water and alkali hydroxide solution, does not attack mercury or brass. Pure SO<sub>2</sub>ClF does not attack glass. M.p. -124.7°C, b.p. 7.1°C; d. (liq.) (0°C) 1.623.

REFERENCES:

I. H. S. Booth and V. Hermann. J. Amer. Chem. Soc. <u>58</u>, 63 (1936).
II. German Patent Application I. 53743.

## Sulfuryl Bromide Fluoride

#### SO<sub>2</sub>BrF

 $Br_2 + BrF_3 = 3 BrF$ 159.82 136.91 296.73

 $3 \text{ BrF} + 3 \text{ SO}_2 = 3 \text{ SO}_2 \text{BrF}$ 296.73 192.18 488.94

The rate of SO<sub>2</sub>BrF formation depends upon the rate at which the following equilibrium is established:  $Br_2 + BrF_3 = 3 BrF$ .

Sulfur dioxide (120 g.) is gradually distilled at  $+12^{\circ}$ C into an iron autoclave containing a mixture of 20 ml. of bromine and 21.2 ml. of BrF<sub>3</sub>. After letting stand for several days, during which the autoclave is shaken once daily, the product is distilled from the pressure vessel and collected in a quartz trap at  $-196^{\circ}$ C For purification, the SO<sub>2</sub>BrF is passed through a wash bottle filled with mercury (removal of traces of bromine and BrF<sub>3</sub>), then over NaF (removal of HF), and finally over P<sub>2</sub>O<sub>5</sub> (removal of water). The product is then fractionated and the first cut discarded. There is no residue. The yield is 88%, based on BrF<sub>3</sub>.

Sulfuryl bromide fluoride is stored by melt-sealing in quartz ampoules.

#### PROPERTIES:

Formula weight 162.98. Colorless compound, choking odor similar to  $SO_2Cl_2$ , thermally stable, reacts slowly with glass at room temperature, unreactive with quartz. Reacts vigorously with water (hydrolysis). On contact with moist air, it acquires a slightly reddish color due to liberation of bromine.

M.p.  $-86^{\circ}$ C, b.p.  $40^{\circ}$ C; d. (liq.) ( $0^{\circ}$ C)2.17, d. (solid) ( $-183^{\circ}$ C)3.16.

REFERENCES:

- O. Ruff and H. Jonas (in collaboration with W. Kwasnik). Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 193.
- H. Jonas, Z. anorg. allg. Chem. 265, 273 (1951).

### Fluorosulfonic Acid

#### HSO<sub>3</sub>F

I.

Dried, powdered  $\text{KHF}_2$  (20 g.) is added with stirring and in small portions to 40 ml. of fuming sulfuric acid (about 60% SO<sub>3</sub>) in a platinum or aluminum dish well cooled with ice-salt mixture. A viscous mass is obtained, which fumes in air. It is then slowly heated to 100°C to drive off unreacted SO<sub>3</sub> and HF.

The fluorosulfonic acid is then distilled in a glass apparatus with ground glass joints by gradual heating to 250°C. The acid is completely pure after a double distillation. The yield is 85%.

II.

$$SO_3 + HF = HSO_3F$$
  
80.06 20 100.07

By means of a capillary made of type 304 stainless steel and dipping below the surface of the liquid, 200 g. of HF is added to 800 g. of  $SO_3$ , kept at 30-35°C in an aluminum vessel. The absorption of HF is rapid but not explosive. The mixture is then heated to 100°C to drive off the excess  $SO_3$  and HF.

The product is distilled twice in an aluminum apparatus.

III.  $HSO_{3}Cl + HF = HSO_{3}F + HCl$ 116.53 20 100.06 36.47

A silver distillation flask equipped with a silver dropping funnel is placed in an ice-salt bath. Anhydrous HF (50 g.) is distilled into the flask through the side arm. A copper drying tube filled with KF is then attached to the side arm, in order to absorb the entrained HF. Then  $HSO_3Cl$  is introduced from the dropping funnel into the flask. The reaction starts immediately and a uniform stream of HCl is given off. After completion of the reaction the excess HF and HCl are removed in a stream of dry air, while slowly heating to  $110^{\circ}C$ . The residue left in the flask is chlorine-free  $HSO_3F$ .

When very pure,  $HSO_3F$  may be stored by sealing into glass ampoules. Otherwise, it should be stored in aluminum vessels.

PROPERTIES:

Colorless liquid, completely stable up to  $900^{\circ}$ C. Reacts explosively with water. Fumes in air. At room temperature does not attack S, C, Se, Te, Pb, Ag, Cu, Zn, Fe, Cr or Mn, but does react with Sn with mild evolution of gas. Mercury is also slightly attacked. Rubber, cork and sealing wax are rapidly destroyed. Vigorously attacks S, Pb, Sn and Hg at higher temperatures. Reacts exothermically with acetone to give a dark red-brown color (color test for fluorosulfonic acid). Reacts with benzene and chloroform, splitting off HF. Ether reacts exothermically and with effervescence to form the ethyl ester. If pure, does not attack glass.

M.p. -87°C, b.p. 163°C; d. (liq.) (18°C) 1.740.

#### REFERENCES:

I. J. Meyer and G. Schramm. Z. anorg. allg. Chem. 206, 25 (1932). II. German Patent Application I 52953 IV b/12 i, August 6, 1935. III. H. Weichert. Z. anorg. allg. Chem. <u>261</u>, 310 (1950).

### **Potassium Fluorosulfinate**

**KSO₂**F

 $KF + SO_2 = KSO_2F$ 58.10 64.06 122.16

One kilogram of anhydrous, finely divided KF is slowly stirred at room temperature for five days with 2 kg. of liquid SO<sub>2</sub> in a 4-liter, agitated iron autoclave. Following that, the excess  $SO_2$  is flushed out; about 2 kg. of 95%  $KSO_2F$  is obtained. The procedure may be altered in the following way: the KF is placed in the autoclave, a steel cylinder containing liquid  $SO_2$  is connected via a capillary, the air in the autoclave is displaced by  $SO_2$ , and the outlet valve of the autoclave is closed. When the agitator is started, there is vigorous absorption of the  $SO_2$  by the KF. After about 1 kg. of  $SO_2$  has been taken up, the rate of absorption begins to fall off.

Potassium fluorosulfinate may be used as "activated potassium fluoride." It reacts with many inorganic and organic acid halides to give the respective fluorides and may therefore often be used as a fluorinating agent in place of anhydrous HF (e.g., preparation of SOF<sub>2</sub>, PF<sub>3</sub>, POF<sub>3</sub>, AsF<sub>3</sub>, C<sub>6</sub>H<sub>5</sub>COF).

PROPERTIES:

Colorless solid, decomposes at 170-180°C. Solubility in liquid  $SO_2$  (0°C) 3.85 mg./100 g. Dissolves in water with hydrolysis. Forms sulfuryl fluoride with Cl<sub>2</sub>, Br<sub>2</sub> or F<sub>2</sub>.

REFERENCE:

F. Seel and L. Riehl. Z. anorg. allg. Chem. 282, 293 (1955).

#### Selenium Hexafluoride

SeF<sub>6</sub>

 $Se + 3F_2 = SeF_6$ 78.96 114 192.96

Selenium is fluorinated in the apparatus described for  $SF_6$  (p. 169), which in this case may be made entirely of glass. The Se ignites in the fluorine stream without external heating. The reaction tube must be cooled from time to time. The product that accumulates in the condensation trap at  $-196^{\circ}$ C is then passed through a fritted wash bottle containing 10% aqueous KOH and is dried over  $P_2O_5$ . Finally, the SeF<sub>6</sub> is completely purified by fractionation.

Selenium hexafluoride may be stored in glass flasks or in a gasometer over water.

PROPERTIES:

Colorless gas, thermally very stable. Does not corrode glass, attacks mercury slightly. When inhaled, causes breathing difficulties and heart seizures. M.p.  $-34.8^{\circ}$ C, subl. t.  $-46.6^{\circ}$ C, t<sub>Cr</sub>  $72^{\circ}$ C; d.(liq.)( $-10^{\circ}$ C) 2.108,d.(solid)( $-195^{\circ}$ C) 3.478.

**REFERENCES:** 

W. Klemm and P. Henkel. Z. anorg. allg. Chem. 207, 74 (1932).

#### Selenium Tetrafluoride

SeF<sub>4</sub>

 $Se + 2F_2 = SeF_4$ 78.96 76 154.96

Selenium, dried at  $200^{\circ}$ C, is spread in a shallow layer in a large-diameter reaction vessel of glass or quartz. The vessel is ice cooled. A fluorine-nitrogen mixture (1 : 1 ratio) is passed over the solid at a rate of 1 liter/hour. Efficient cooling and careful fluorination are important, since otherwise SeF<sub>6</sub> is formed. Finally, the liquid product is vacuum distilled.

PROPERTIES:

Colorless liquid, miscible with sulfuric acid, alcohol, ether and  $IF_5$ . Dissolves NaF, KF, RbF, CsF and TlF with formation of the complex MSeF<sub>5</sub>. Water decomposes SeF<sub>4</sub> vigorously. Forms HgSeF<sub>4</sub> when refluxed with mercury for several hours. Slowly attacks Pyrex glass.

M.p. -9.5°Č, b.p. 106°C; d<sup>25</sup> 2.72.

**REFERENCES:** 

- E. E. Aynsler, R. D. Peacock and P. L. Robinson. J. Chem. Soc. (London) <u>1952</u>, 1231.
- R. D. Peacock. J. Chem. Soc. (London) 1953, 3617.

#### Tellurium Hexafluoride

TeF<sub>6</sub>

 $\begin{array}{c} Te + 3 \, F_2 = TeF_6 \\ 127.61 \quad 114 \quad 241.61 \end{array}$ 

Tellurium is fluorinated in the apparatus described for  $SF_6$  (p. 169), which in this case may be entirely of glass. The reaction

is exothermic, but ignition does not usually occur if the reaction is moderated with external cooling.

The product collecting in the trap at  $-196^{\circ}$ C is fractionated. Preliminary washing of the gas with potassium hydroxide solution is not feasible since TeF<sub>6</sub> is hydrolytically cleaved by alkali. Tellurium hexafluoride may be stored in glass flasks.

PROPERTIES:

Colorless gas, unpleasant odor, chemically not quite as inert as  $SeF_6$  and  $SF_6$ . Slowly but completely hydrolyzed by water. Attacks mercury. Causes breathing difficulties and heart seizures. After inhalation, the well known disagreeable odor of tellurium is noticeable.

M.p.  $-37.6^{\circ}$ C, subl. t.  $-38.9^{\circ}$ C, t<sub>cr</sub>  $83^{\circ}$ C; d. (liq.)( $-10^{\circ}$ C)2.499, d. (solid)( $-191^{\circ}$ C) 4.006.

REFERENCES:

W. Klemm and P. Henkel. Z. anorg. allg. Chem. 207, 74 (1932).
D. M. Yost and W. H. Claussen. J. Amer. Chem. Soc. 55, 885 (1933).

### Nitrogen Trifluoride

NF<sub>3</sub>

 $\begin{array}{r} 4\,\mathrm{NH}_{3}\,+\,3\,\mathrm{F}_{2}\,=\,\mathrm{NF}_{3}\,+\,3\,\mathrm{NH}_{4}\mathrm{F}\\ 68.12\,\,\,114.0\,\,\,71.0\,\,\,111.12 \end{array}$ 

Nitrogen trifluoride is made by electrolysis of molten  $NH_4HF_2$ , during which  $NH_3$  is fluorinated by nascent fluorine.

Chlorine-free  $\text{NH}_4\text{HF}_2$ , as dry as possible, is electrolyzed in an electrolytic cell such as that described for the preparation of  $F_2$  (see section on Fluorine). The temperature is maintained at 130-140°C. Acheson graphite is used as the anode. The operating current is 10 amp., resulting in a voltage of 7 to 9 v. The current density at the anode is 0.05 to 0.1 amp./cm<sup>2</sup>. (The current density does not influence the yield.)

The reaction gases from the electrolytic cell are passed through an iron drying tube containing freshly dehydrated KF to remove entrained HF and water. The apparatus ends in an iron drying tube containing KH.

After the start of the electrolysis, only solid, partly colorless, partly violet or blue products  $(N_2O_3, N_2O_3, O_3)$  condense in the traps in the first few hours (or days), depending on the amount of moisture in the electrolyte. Explosions caused by ozone may sometimes occur at the start of the electrolysis, and this must be taken into account. The rate of deposition of solids drops as the electrolysis removes the moisture from the melt, and increasing amounts of colorless, liquid  $NF_3$  begin to appear. However, the yield of  $NF_3$ , based on the current, never rises above 30% of theoretical.

It should be borne in mind that the nature of the products of electrolysis depends strongly on the anode material. Swedish graphite generates only  $N_2$ , whereas aluminum carbide or nickel anodes produce only fluorine gas from the same melt.

American graphite yields a maximum of 30% NF<sub>3</sub>, arc carbon 16%, and carbon welding electrodes 18%. Various carbon anodes must be tried before optimum results are obtained.

The reaction mixture, condensed in the traps as a slurry, is first washed with potassium hydroxide solution to remove acidic components. The product is then fractionated to separate the greater part of the N<sub>g</sub>O. At this stage, the NF<sub>3</sub> becomes a colorless liquid covered with a white layer of solid N<sub>g</sub>O. In order to remove this, the NF<sub>3</sub> must be repeatedly and very carefully fractionated. The last traces of N<sub>g</sub>O may be more conveniently separated by filtering the NF<sub>3</sub> at  $-196^{\circ}$ C on a low-temperature filter. It is completely pure after only one filtration. Finally, the air dissolved in the NF<sub>3</sub> is removed, using an oil pump vacuum for several hours, while the trap with the product is immersed in liquid nitrogen. Purity is best ascertained by molecular weight determination (vapor density measurement).

Nitrogen trifluoride is stored in glass flasks, in a gasometer over water, or under pressure in steel cylinders. It may be used as filling material for vapor pressure thermometers [W. Menzel and F. Mohry, Z. anorg. allg. Chem. 210, 257 (1933)].

#### PROPERTIES:

Colorless, very stable at room temperature. Does not react with water or KOH solution at room temperature unless a spark is discharged. Does not attack glass and mercury. Characteristic decay odor. Not dangerous if pure and when not inhaled in high concentrations. Crude NF<sub>3</sub>, however, has a much more unpleasant effect because of its impurity content. Causes headache, nausea and diarrhea.

M.p.  $-208.5^{\circ}$ C, b.p.  $-129^{\circ}$ C;  $\Delta$ H (formation) +26 kcal; d. (liq.) (-129°C) 1.855.

REFERENCES:

O. Ruff, F. Luft and J. Fischer. Z. anorg. allg. Chem. <u>172</u>, 417 (1928).

- O. Ruff. Z. anorg. allg. Chem. <u>197</u>, 273 (1931).
- O. Ruff and L. Staub. Z. anorg. allg. Chem. <u>198</u>, 32 (1931).
  W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 204.

### Ammonium Fluoride

#### NH₄F

I.

 $NH_3 + HF = NH_4F$ 17.03 20 37.03

Excess gaseous ammonia is added to an ice-cooled platinum or lead dish containing 40% hydrofluoric acid. The ammonium fluoride that separates is suction filtered.

II.  $NH_4Cl + N_8F = NH_4F + N_8Cl$ 53.49 42.0 37.03 58.46

A mixture of 1 part  $NH_4Cl$  and  $2^{1}/_4$  parts NaF is gently heated in a platinum crucible. Ammonium fluoride sublimes and is collected on the cooled crucible lid in the form of very small, prismatic, very pure crystals.

Solid ammonium fluoride cannot be obtained by evaporation of an NH<sub>4</sub>F solution, since NH<sub>3</sub> splits off and NH<sub>4</sub>HF<sub>2</sub> is formed. Ammonium fluoride is stored in iron vessels.

PROPERTIES:

White, deliquescent, crystalline flakes or needles, very soluble in water; decomposes on heating into  $NH_3$  and HF. Attacks glass. Solubility in water at  $0^{\circ}C$ : 100 g./100 ml.

d. 1.015. Structure: hexagonal (wurtzite).

REFERENCE:

J. J. Berzelius. Lehrbuch der Chemie [Textbook of Chemistry], 5th ed., Vol. III, p. 282.

### Ammonium Hydrogen Fluoride

#### $\mathbf{NH_4F}\boldsymbol{\cdot}\mathbf{HF}$

 $NH_3 + 2 HF = NH_4F \cdot HF$ 17.03 40.02 57.04

Gaseous ammonia is added to a platinum or lead dish containing 40% hydrofluoric acid until the color of Congo paper changes to brown. The solution is then cooled with ice, whereupon  $NH_4HF_2$  separates out. It is filtered off and dried by suction.

PROPERTIES:

White, rhombic crystals. M.p. 124.6°C; d 1.503.

REFERENCE:

O. Hassel and N. Luzanski. Z. Kristallogr. A 83, 440 (1932).

# Nitrosyl Fluoride

#### NOF

I.

NOBF <sub>4</sub>	+ NaF	= NOF	- -	NaBF <sub>4</sub>
116.83	42	49.01		109.82

A nickel tube closed at one end serves as the reaction vessel. The tube end projecting out of the furnace is wrapped with a lead cooling coil through which cooling water flows (see Fig. 120). Two



Fig. 120. Preparation of nitrosyl fluoride.

quartz liquid nitrogen-cooled condensation traps are connected to the outlet of the nickel tube. The reaction apparatus also includes a spiral quartz manometer and an air inlet protected by a  $P_2O_5$  tube (stopcock lubricant: vaseline with graphite). The apparatus is connected to a mercury diffusion pump. A liquid nitrogen-cooled trap is inserted ahead of the pump to retain undesirable acid fumes.

The nickel tube is filled with  $\text{NOBF}_4$  (as pure as possible) and an excess of dry NaF. This must be done under dry nitrogen. The quartz traps are then connected. The ground joints are not greased

184

but made airtight from outside with picein. Following this, the mercury pump is turned on and the system is brought down to a pressure of 0.01 mm. The furnace is then gradually heated to  $300^{\circ}$ C. The reaction begins at  $100^{\circ}$ C, at  $200^{\circ}$ C the rate is already considerable, and at  $300^{\circ}$ C it proceeds very vigorously. Blue NOF condenses in the first and, to some extent, also in the second trap. About 10 ml. of crude NOF is obtained from 30 g. of NOBF 4.

After completion of the reaction, the NOF is fractionated and collected in quartz receptacles. However, only rarely can it be made completely colorless, and in most cases it retains a slightly bluish color.

This procedure has the advantage of not requiring the use of elemental fluorine.

п.

$$\begin{array}{ccc} 2 \, \text{NO} \, + \, \mathbf{F_2} = 2 \, \text{NOF} \\ 60.02 & 38 & 98.02 \end{array}$$

Dry NO at a rate of 55 ml./min. and fluorine at a rate of 26 ml./min. are mixed in the apparatus described for the preparation of  $SOF_4(p. 172)$ . There should always be excess NO. The reaction tube need not be heated, since the reaction between the two gases is slightly exothermic. The quartz traps are maintained at  $-120^{\circ}C$  and the crude NOF condenses as a blue liquid.

After the fluorination, the crude NOF is fractionated several times and collected in quartz vessels. The first cuts consist of NO dissolved in liquid NOF and some  $SiF_{4*}$ . The residue is  $N_2O_3$ .

Nitrosyl fluoride is best stored in quartz ampoules cooled with liquid nitrogen.

SYNONYM:

Nitrogen oxyfluoride.

PROPERTIES:

Colorless if pure, but owing to impurities the liquid often has a bluish hue. Dissolves in water, yielding a blue color, but decomposes rapidly into NO and HNO<sub>3</sub>. Reacts vigorously with glass but less readily with quartz.

M.p.  $-132.5^{\circ}$ C, b.p.  $-59.9^{\circ}$ C; d.(liq.)( $-59^{\circ}$ C) 1.326, d.(solid) 1.719.

REFERENCES:

- I. G. Balz and E. Mailänder. Z. anorg. allg. Chem. 217, 166 (1934).
- II. O. Ruff, W. Menzel and W. Neumann. Z. anorg. allg. Chem. 208, 293 (1932).

### Nitrososulfuryl Fluoride

#### FSO<sub>2</sub>NO

 $\begin{array}{rrr} \text{NOF} + \text{SO}_2 = \text{FSO}_2\text{NO} \\ 49.01 & 64 & 113.01 \end{array}$ 

Sufficient SO<sub>2</sub> is condensed over NOF so that, after melting, a suspension of FSO<sub>2</sub>NO in liquid SO<sub>2</sub> is obtained. The nitrososulfuryl fluoride is separated from the mixture by partial evaporation and condensation.

The compound is synthetically useful since itacts as "stabilized NOF." It can easily be prepared in pure form, and may be stored in glass containers in a Dry Ice chest. Polyethylene has proven to be an ideal container material for reactions with FSO<sub>2</sub>NO.

PROPERTIES:

Colorless, scintillating crystalline flakes, easily sublimed. M.p. (under a pressure of 3100 mm. abs.) 8°C. At 19°C, the compound is 70% decomposed to its constituents.

**REFERENCE:** 

F. Seel and H. Massat. Z. anorg. allg. Chem. 280, 186 (1955).

#### Nitryl Fluoride

NO<sub>2</sub>F

I.

 $2 \operatorname{NO}_2 + \operatorname{F}_2 = 2 \operatorname{NO}_2 \operatorname{F}_{92.0}_{38.0} \operatorname{I30.0}_{130.0}$ 

Nitrogen dioxide is prepared from NO and excess oxygen. It is dried over  $P_2O_5$ , condensed on an ice-salt mixture, and distilled once in a stream of oxygen.

A 25-ml./min. stream of fluorine and an 18.6-ml./min. stream of NO<sub>2</sub> are mixed in the apparatus described for the preparation of SOF<sub>4</sub> (p.172). The NO<sub>2</sub> from the storage vessel is introduced into the apparatus in a stream of oxygen. The reaction is slightly exothermic. The crude NO<sub>2</sub>F collected in the condensation traps at  $-120^{\circ}$ C is colorless. It is purified by fractional distillation and collected in quartz containers. The first cut consists of F<sub>2</sub> and some SiF<sub>4</sub>; there is virtually no residue. II.

 $N_{2}O_{5} + BF_{3} + HF = NO_{2}[BF_{4}] + HNO_{3}$ 108.0 67.82 20 132.82 63

$$\frac{NO_{2}[BF_{4}] + NaF}{132.82} + \frac{NaF}{42} = \frac{NO_{2}F}{65.0} + \frac{NaBF_{4}}{109.82}$$

The following procedure is available if elemental fluorine cannot be used: stoichiometric quantities of anhydrous HF and BF<sub>3</sub> are added to a solution of  $N_2O_5$  in nitromethane. Nitryl fluoroborate precipitates out of the nitromethane solution. The crystals are filtered with exclusion of moisture and then heated with NaF to  $240^{\circ}$ C in a platinum or nickel vessel, analogous to procedure I in the preparation of NOF (p. 184).

Since  $N_2O_5$  decomposes easily, there is an advantage in stabilizing it. Thus,  $N_2O_5$  and  $BF_3$  are first allowed to react to give the compound  $N_2O_5 \cdot BF_3$ , which is stable at room temperature in a dry atmosphere. This compound may then be dissolved in nitromethane and allowed to react with anhydrous HF.

Nitryl fluoride is best stored in quartz ampoules cooled with liquid nitrogen.

PROPERTIES:

Formula weight 65.0. Colorless gas and liquid, white solid. Has a penetrating odor and strongly attacks the mucous membranes. Hydrolyzed by water. Absorbs mercury without leaving a residue. Reacts with most metals and nonmetals, and vigorously with alcohol, ether, benzene and chloroform.

M.p.  $-166.0^{\circ}$ C, b.p.  $-72.4^{\circ}$ C; d. (liq.) ( $-72^{\circ}$ C) 1.796, d. (solid) 1.924.

**REFERENCES:** 

- I. O. Ruff, W. Menzel and W. Neumann. Z. anorg. allg. Chem. 208, 298 (1932).
- II. M. Schmeisser and S. Elischer. Z. Naturforschg. <u>7</u> b, 583 (1952).

#### **Fluorine Nitrate**

NO<sub>3</sub>F

$$\frac{\text{HNO}_3 + F_2}{63.01} = \frac{\text{NO}_3 F}{38} + \frac{\text{HF}}{81.01} = \frac{1}{20}$$

Fluorine, held ready in a quartz trap immersed in liquid nitrogen, is aspirated into the apparatus through a copper diaphragm valve



Fig. 121. Preparation of fluorine nitrate.

(see Fig. 121). The fluorine can, of course, also be taken directly from a steel cylinder. The quartz reaction vessel contains 100% HNO<sub>3</sub>, which has been prepared by distilling a mixture of fuming nitric acid and concentrated sulfuric acid at 20 mm. at room temperature. It is protected from decomposition by storage in liquid nitrogen. A quartz drying tube containing freshly dehydrated KF for removal of HF is connected to the reaction vessel. This is followed by a train of three quartz traps, which are cooled with liquid nitrogen. The first two serve as receivers; the third merely to keep out moisture. To help regulate the vacuum, a glass manometer is included. The system ends in a glass stopcock. A metal aspirator produces the vacuum.

The fluorine rate is controlled by the diaphragm valve so that two or three gas bubbles per second are delivered to the reaction vessel at a pressure of 20 mm. At this flow rate there is no noticeable rise in the temperature. In order to prevent plugging of the condensation traps, the coolant should be removed for a short while every 15 minutes, so that the accumulated solid  $NO_3F$  may melt and collect at the bottom.

Care should be taken to use completely grease-free apparatus for this reaction, since otherwise there is a risk of explosion.

After completion of the reaction the  $NO_3F$  is purified by fractional distillation in a quartz column at a pressure of 100 mm. The fraction boiling at  $-79^{\circ}C$  (99 mm.) is pure  $NO_3F$ . The first cut consists of SiF<sub>4</sub>; the residue, chiefly of H<sub>2</sub>SiF<sub>6</sub> and HF.

Fluorine nitrate may be stored by sealing it under vacuum (less than 0.1 mm.) into quartz or glass ampoules, but caution is necessary, since explosions do sometimes occur. The safest way to store the compound is to keep the ampoules in liquid nitrogen.

#### PROPERTIES:

Colorless; has a repellent, musty odor; causes severe irritation of the respiratory tract, headaches and breathing difficulties, which persist for several days. Liquid  $NO_3F$  explodes when vigorously shaken. Hydrolyzed by water to  $OF_2$ ,  $O_2$ , HF and  $HNO_3$ . Apparently quite soluble in acetone. Explodes immediately on contact with alcohol, ether and aniline.

M.p.  $-175^{\circ}$ C, b.p.  $-45.9^{\circ}$ C; d.(liq.)( $-45.9^{\circ}$ C) 1.507, d.(solid) ( $-193.2^{\circ}$ C) 1.951.

**REFERENCE:** 

O. Ruff and W. Kwasnik. Angew. Chem. 48, 238 (1935).

# Phosphorus (III) Fluoride

PF<sub>3</sub>

 $PCl_3 + 3 HF = PF_3 + 3 HCl$ 137.35 60 87.98 109.38

A 70-cm.-long, 4-cm. diameter quartz or iron tube closed at one end serves as the reaction vessel (Fig. 122). An iron capillary, reaching nearly to the bottom of the vessel, is inserted through the rubber stopper. An iron or quartz reflux condenser is attached above the reaction tube. From there a connection leads to the condensation trap (quartz or glass), which is cooled with liquid nitrogen.

The reaction vessel is filled with  $PCl_3$ , and gaseous HF is added. At 50 to 60°C the reaction proceeds smoothly. After



Fig. 122. Preparation of phosphorus trifluoride.

completion of the reaction the mixture of  $PF_3$  and HCl collecting in the condensation trap is rapidly passed through wash bottles containing ice-cold water (where the HCl is absorbed), dried over  $P_2O_5$ , and distilled. The yield is greater than 90%, based on PCl<sub>3</sub>.

For larger-scale preparations, the apparatus shown on p. 171  $(SOF_2)$  may be used.

Phosphorus (III) fluoride is stored in steel cylinders or in glass flasks. It may also be stored in a gasometer over mercury.

PROPERTIES:

Colorless gas, does not fume in air, almost odorless, poisonous (causes difficulty in breathing, chest pains, nausea). Only slowly hydrolyzed by water; does not attack glass.

M.p. -151.5°C, b.p. -101.8°C, t<sub>cr</sub> -2°C, p<sub>cr</sub> 42.7 atm.

REFERENCE:

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 213.

## Phosphorus (V) Fluoride

PF5

3 PCl <sub>3</sub>	+	5 AsF <sub>3</sub>	=	$3\mathrm{PF}_5$	+	5 AsCl <sub>3</sub>
624.8		659.6		377.9		906.8

Arsenic trifluoride from a dropping funnel is allowed to flow in drops into a glass flask containing  $PCl_{5}$  (see Fig. 123). A trap immersed in liquid nitrogen (the  $PF_{5}$  receiver) is connected by a ground glass joint to the outlet of the flask. At the system outlet there is a drying tube filled with anhydrous KF for exclusion of atmospheric moisture.

The reaction starts without warming. Phosphorus (V) fluoride contaminated with  $AsF_3$  accumulates as a white solid in the trap. After completion of the conversion the crude  $PF_5$  is purified by fractionation.

The product may be stored under pressure in steel cylinders or in glass flasks.



Fig. 123. Preparation of phosphorus pentafluoride. PROPERTIES:

Formula weight 125.98. Colorless gas, strongly fuming in air; attacks the skin and lungs. Rapidly hydrolyzed by water. Does not attack dry glass at room temperature.

M.p.  $-83^{\circ}$ C, b.p.  $-75^{\circ}$ C.

**REFERENCE:** 

O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin 1920, p. 29.

#### **Phosphorus Dichloride Fluoride**

PCl<sub>2</sub>F

 $\begin{array}{rrrr} PCl_3 \ + \ SbF_3 \ = \ PCl_2F \ + \ SbClF_2 \\ 137.35 \ 178.76 \ 120.90 \ 195.21 \end{array}$ 

A two-liter, three-neck, round-bottom flask is used as the reaction vessel (Fig. 124). A vacuum-tight stirring arrangement



Fig. 124. Preparation of phosphorus dichloride fluoride.

is inserted into the middle neck. The first neck is for introducing  $SbF_3$ , either by means of a worm-screw conveyor arrangement or, more simply, by a flexible rubber tube from a round-bottom flask. The third neck supports a 1-m.-long glass column to which
a partial condenser is affixed. The reaction gases flow from the condenser into a trap immersed in liquid nitrogen. Next, there is a drying tube filled with freshly dehydrated KF, followed by a stopcock, which allows the system to be separated from the manometer and the aspirator.

The reaction flask is filled with 130 g. of PCl<sub>3</sub> and 2 g. of PCl<sub>5</sub> (catalyst). The system is then evacuated to 250 mm., and this pressure is maintained during the entire synthesis. The partial condenser is fed with flowing water. Then 175 g. of dry, powdered  $SbF_3$  is gradually added to the reaction vessel over a period of three hours. By cooling or heating, as necessary, a constant temperature of about 40°C is maintained.

The crude PCLF is collected in the trap. After completion of the reaction, it is fractionated. The yield is 60%.

Phosphorus dichloride fluoride is best stored by sealing into glass ampoules at  $-78^{\circ}$ C; if necessary, it may be kept for short periods in steel cylinders at room temperature.

SYNONYM:

Dichlorofluorophosphine.

PROPERTIES:

Colorless gas, unstable at room temperature. Does not fume in air; hydrolyzed by water; absorbed completely by sodium hydroxide solution with evolution of heat.

M.p. -144°C, b.p. 13.85°C; d 1.507.

REFERENCE:

H. S. Booth and A. R. Bozart. J. Amer. Chem. Soc. 61, 2927 (1939).

## Phosphorus Dichloride Trifluoride

 $PCl_2F_3$ 

 $PF_3 + Cl_2 = PCl_2F_3$ 88.02 70.92 158.94

Equal metered volumes of  $PF_3$  and  $Cl_2$  flow into a 1-m.-long quartz tube. This serves as the reaction vessel (the apparatus is similar to that shown in Fig. 119), where the exothermic addition of  $Cl_2$  to  $PF_3$  takes place. The quartz tube is connected via a ground joint to a quartz trap immersed in liquid nitrogen. A drying tube with KF is attached to the trap to exclude atmospheric moisture. The product accumulating in the condensation trap is fractionated after completion of the reaction.

Phosphorus dichloride trifluoride may be stored inglass flasks.

PROPERTIES:

Colorless gas, very pungent odor, attacks the respiratory organs, forms a thick white fog in air. Disproportionates on heating to  $200^{\circ}$ C. Excess water absorbs PCl<sub>2</sub>F<sub>3</sub> without residue, producing H<sub>3</sub>PO<sub>4</sub>, HF and HCl. With little water, POF<sub>3</sub> and HCl are formed and a rise in volume is observed. Alcohol solvolyzes the gas.

REFERENCES:

C. Poulenc. Compt. Rend. hebd. Séances Acad. Sci. <u>113</u>, 75 (1891). V. Schomaker and J. B. Hatscher. J. Amer. Chem. Soc. <u>60</u>, 1837 (1938).

## **Phosphorus Oxide Trifluoride**

#### POF<sub>3</sub>

 $\begin{array}{l} \text{POCl}_3 + 3 \text{ HF} = \text{POF}_3 + 3 \text{ HCl} \\ 153.35 & 60 & 104.0 & 109.38 \end{array}$ 

Gaseous HF is introduced into the apparatus described for  $PF_3$  (p. 189), containing POCl<sub>3</sub> at 65°C. Antimony pentachloride (5 wt.%) is added as catalyst.

The product ( $POF_3 + 3$  HCl) accumulating in the condensation trap is separated by repeated fractionation. The yield is greater than 90%, based on  $POCl_3$ .

Phosphorus oxide trifluoride is stored in glass flasks or steel cylinders.

PROPERTIES:

Colorless, pungent gas; fumes slightly in air. M.p. -39.4°C, subl. t. -39.8°C, t<sub>cr</sub> 73.3°C, p<sub>cr</sub> 41.8 atm.

REFERENCE:

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 213.

## Tetrachlorophosphonium Hexafluorophosphate (V)

## PCl<sub>4</sub> · PF<sub>6</sub>

 $\begin{array}{rrrr} P_2 Cl_{10} \ + \ 2 \ A_5 F_3 \ = \ P Cl_4 \cdot P F_6 \ + \ 2 \ A_5 Cl_3 \\ 416.53 \ & 263.82 \ & 317.89 \ & 362.46 \end{array}$ 

Phosphorus pentachloride (46 g.) is dissolved in 300 ml. of AsCl<sub>3</sub>. The solution is stirred and slightly cooled while 29.8 g. of

 $AsF_3$  is added dropwise. The product  $(PCl_4 \cdot PF_6)$  precipitates as fine white crystals. The end point of the reaction is indicated by the formation of  $PF_5$  (thick white fog). The precipitate is filtered with exclusion of moisture on a fritted glass filter, washed with  $AsCl_3$ , and freed of adhering  $AsCl_3$  in a stream of dry air. The yield is 35 g. (quantitative).

The compound is a convenient starting material for the preparation of hexafluorophosphates (hydrolysis with the respective hydroxides; see p. 196 under  $\text{KPF}_6$ ) and of  $\text{PF}_5$  (thermal decomposition at 80°C).

PROPERTIES:

White, hygroscopic salt, very slightly soluble in AsCl<sub>3</sub>. M.p. 160°C (partial dec.), subl. t. 135°C (partial dec.).

**REFERENCE:** 

L. Kolditz. Z. anorg. allg. Chem. 284, 144 (1956).

# **Phosphonitrilic Fluorides**

 $(PNF_2)_3$ ,  $(PNF_2)_4$ 

 $(PNCl_2)_3 + 6 KSO_2F = (PNF_2)_3 + 6 KCl + 6 SO_2$ 347.7 732,96 248.94 447.36 384.36

 $(PNCl_2)_4 + 8 \text{ } \text{KSO}_2\text{F} = (PNF_2)_4 + 8 \text{ } \text{KCl} + 8 \text{ } \text{SO}_2 \\ 463.6 \quad 977.28 \quad 331.92 \quad 596.48 \quad 512.48 \\ \end{cases}$ 

Powdered trimeric or tetrameric phosphonitrilic chloride is reacted with potassium fluorosulfinate at 120 to  $150^{\circ}$ C. The degree of polymerization is not altered by the reaction.

PROPERTIES:

Both phosphonitrilic fluorides are solid, colorless, volatile substances at room temperature. They are thermally stable up to  $300^{\circ}$ C. The trimer boils at  $51.8^{\circ}$ C and crystallizes in monoclinic prisms. Triple point 27.1°C. It polymerizes to a rubbery form by heating for 15 hours at  $350^{\circ}$ C. The tetramer boils at  $89.7^{\circ}$ C and forms triclinic-pinacoidal crystals. Triple point  $30.4^{\circ}$ C.

REFERENCE:

F. Seel and J. Langer. Angew. Chem. 68, 461 (1956).

194

# Ammonium Hexafluorophosphate (V) NH<sub>4</sub>PF<sub>6</sub>

Ι.

 $PCl_5 + 6 NH_4F = NH_4PF_6 + 5 NH_4Cl$ 208.31 222.24 163.06 267.45

A mixture of 9.4 g. of PCl<sub>5</sub> and 11.6 g. of dry  $NH_4F$  is prepared by shaking in a test tube. The open test tube is fastened to a stand in near-horizontal position and the mixture is heated with a small flame near the open end until the reaction starts (use hood, goggles). The reaction then progresses spontaneously until the bottom of the test tube is reached and fuming, heavier-than-air vapors are liberated. After cooling, the solid mass formed in the test tube is dissolved in two liters of water. An acetic acid solution (100 ml.) containing 9 g. of nitron is slowly poured with stirring into the solution, so that nitron hexafluorophosphate precipitates out.

After cooling with ice for two hours the salt is filtered, washed several times with a little ice-cold water and, while still moist, shaken with chloroform and 25% ammonia solution in a separatory funnel. After the nitron has thus been removed, the aqueous solution is evaporated to dryness in a platinum dish on a water bath. The yield is 4 g. of NH<sub>4</sub>PF<sub>e</sub>.

For purification the salt is dissolved in a small amount of water, filtered and reevaporated in a platinum dish, but only until a wet mass of crystals appears. This is spread in a clav dish and left to dry in the air.

II.

 $(NPCl_2)_n + 6 n HF = n NH_4PF_6 + 2 n HCl$ 236.06 120 163.06 73

Phosphonitrilic chloride is wetted with hydrofluoric acid in a platinum dish. An exothermic reaction occurs. The mixture is evaporated to dryness on a water bath. The yield is quantitative. In comparison with method I, this procedure has the advantage that it leads directly to a pure product. The purification using nitron hexafluorophosphate is therefore omitted.

This compound may be used for the preparation of many salts of hexafluorophosphoric acid.

PROPERTIES:

Colorless, mostly square, rarely rectangular flakes or thick plates, readily soluble in water; also soluble in acetone, methyl and ethyl alcohols; decomposes on heating to a relatively high temperature without prior melting. Does not attack glass at room temperature. Slowly hydrolyzed by boiling with strong acids. d<sup>1</sup>/<sub>4</sub> 2.180. Solubility in water at 20°C: 74.8 g./100 ml. Structure:

cubic.

**REFERENCES:** 

- I. W. Lange and E. Müller. Ber. dtsch. chem. Ges. <u>63</u>, 1063 (1930); W. Lange and G. v. Krueger. Ber. dtsch. chem. Ges. 65, 1265 (1932).
- II. H. Bode and H. Clausen. Z. anorg. allg. Chem. 265, 229 (1951).

## Ammonium Difluorophosphate (V)

## $\mathbf{NH_4PO_2F_2}$

Phosphorus pentoxide (23.5 g.) is heated with 185 g. of  $NH_4F$ in a 300-ml. nickel or copper crucible until the reaction starts. It progresses by itself, but the mixture should be well stirred. After cooling, the mass is pulverized and boiled in a glass flask with 600 ml. of absolute alcohol. The mixture is filtered hot through a fluted filter; the filtrate is immediately cooled and neutralized with ammoniacal alcohol. Ammonium difluorophosphate (V) (8.3 g.) separates out and is removed by filtration. The filtrate is evaporated to dryness in a platinum dish on a water bath. The yield is 11.6 g. (70% of theoretical) of crude salt. This  $NH_4PO_3F_2$  is still contaminated with  $NH_4F$  but is suitable for many purposes.

The salt is purified by rapid recrystallization from 6 ml. of hot water and drying over  $H_2SO_4$ . The yield is 3.2 g. (20% of theoretical) of analytically pure salt. It is stored in a glass container with exclusion of atmospheric moisture.

PROPERTIES:

Formula weight 134.1. Colorless; gives a neutral reaction in water at first but hydrolyzes with time. Readily soluble in water, ethyl and methyl alcohols and acetone.

M.p. 213°C without decomposition. Structure: rhombic.

REFERENCE:

W. Lange. Ber. dtsch. chem. Ges. 62, 790 (1929).

## Potassium Hexafluorophosphate (V)

KPF6

 $\begin{array}{rrr} PCl_4 \cdot PF_6 + 7 \text{ KOH} = \text{KPF}_6 + \text{K}_2 \text{HPO}_4 + 4 \text{ KCl} + 3 \text{ H}_2 \text{O} \\ 317.89 & 392.51 & 183.98 & 174.18 & 298.24 & 54 \end{array}$ 

Hydrolysis of 0.78 g. of  $PCl_4 \cdot PF_6$  in 20 ml. of 11N potassium hydroxide yields a solution, which is concentrated under vacuum at

 $45^{\circ}C$  to 3 ml. The crystalline precipitate that separates out is filtered, washed with alcohol, and dried.

PROPERTIES:

Square and rectangular thick plates, face-centered cubic lattice. Melts at red heat with partial decomposition. On heating with solid NaOH, a vigorous reaction starts above  $400^{\circ}$ C, giving the fluoride and the phosphate.

REFERENCE:

L. Kolditz. Z. anorg. allg. Chem. 284, 144 (1956).

## Arsenic (III) Fluoride

#### AsF<sub>3</sub>

I.

 $\begin{array}{r} As_2O_3 + 6 HF = 2 AsF_3 + 3 H_2O \\ 197.80 & 120.06 & 263.82 & 54.04 \end{array}$ 

At a bath temperature of  $140^{\circ}$ C, anhydrous HF is fed into As<sub>2</sub>O<sub>3</sub> contained in a distillation apparatus made entirely of iron (see Fig. 125). The steel cylinder with the HF is immersed in a water bath at +35°C. The AsF<sub>3</sub> distilling off is condensed in a brine-cooled condenser maintained at -18°C. The rate of HF addition is regulated in such a way that a smooth stream of liquid AsF<sub>3</sub> flows out of the condenser.

After shutting off the flow of HF, the reaction vessel is removed, and 10% of  $H_2SO_4$  (by volume) is added to the crude AsF<sub>3</sub>. The vessel is then used as a distillation flask and the product is distilled. The main fraction (between 50°C and about 85°C) is AsF<sub>3</sub>. The yield is 80%, based on As<sub>2</sub>O<sub>3</sub>. Six kilograms can readily be prepared in a day.

II.  $2 A_{s_2O_3} + 6 HSO_3F = 2 A_sF_3 + SO_3 + 3 H_2SO_4 + A_{s_2O}(SO_4)_2$ 395.64 600.42 263.82 80 294.24 358

A mixture of 144 g. of  $As_2O_3$  and 247 g. of  $HSO_3F$  (40% excess) is prepared in a glass round-bottom flask provided with a groundglass joint. The latter supports a large-diameter, air-cooled reflux condenser. An inclined condenser and an ice-cooled receiver are attached to the reflux condenser. A noticeable temperature rise results from the mixing. While an air flow into the flask (through the reflux condenser) is induced by suction in order to retain the  $HSO_3F$ , the flask is heated on an open flame. In less than 1.5 hours, about 60 g. of  $AsF_3$  distills over at 58 to 62°C. The yield is 78%, based on  $As_2O_3$ .



Fig. 125. Preparation of arsenic (III) fluoride.

Arsenic (III) fluoride is stored in iron vessels.

PROPERTIES:

Formula weight 131.9. Colorless, mobile, very poisonous liquid. Fumes in air, attacks glass. Decomposed by water as soon as the stoichiometric ratio is reached. Soluble in alcohol, ether and benzene.

M.p.  $-8.5^{\circ}$ C, b.p.  $+63^{\circ}$ C; d. (liq.) ( $15^{\circ}$ C) 2.73.

**REFERENCES:** 

I. W. Kwasnik. Not yet published.

II. A. Engelbrecht, A. Aignesberger and E. Hayek. Mh. Chem. 86, 470 (1955).

## Arsenic (V) Fluoride

AsF5

$$\begin{array}{r} 2\,\mathrm{As}\,+\,5\,\mathrm{F_2}\,=\,2\,\mathrm{AsF_5}\\ 149.82\,\,\,190.0\,\,\,\,339.82 \end{array}$$

Arsenic is fluorinated in a nickel or alumina boat, using the apparatus described for  $SF_6$  (p. 169). The product condensed in the traps is distilled several times in a quartz apparatus.

Arsenic (V) fluoride is stored in steel cylinders.

PROPERTIES:

Formula weight 169.91. Colorless gas. Forms white clouds in moist air. Immediately hydrolyzed by water. Soluble in alcohol, ether and benzene.

M.p. -79.8°C, b.p. -52.9°C; d. (liq.) (-52.8°C) 2.33.

REFERENCE:

O. Ruff, W. Menzel and H. Plaut. Z. anorg. allg. Chem. 206, 61 (1932).

## Antimony (III) Fluoride

#### SbF<sub>3</sub>

 $\begin{array}{rl} Sb_2O_3 \,+\, 6\,HF \,=\, 2\,SbF_3 \,+\, 3\,H_2O \\ 291.52 & 120 & 357.52 & 54.03 \end{array}$ 

I. Antimony (III) oxide is dissolved in excess aqueous hydrofluoric acid and the solution evaporated to dryness on a hot plate.

The product is then distilled in a copper apparatus. The distillation vessel is conical at the top, and a short, large-diameter head is used. The head must be kept sufficiently warm during the distillation to prevent plugging.

II. Gaseous HF is added through a silver capillary tube to  $Sh_2O_3$  contained in a conical vessel made of Mg sheet and covered with an Mg cover; the vessel is heated gently with a gas flame during the addition. When no further HF is absorbed, the heating is increased to evaporate the accumulated H<sub>2</sub>O. The addition of HF and evaporation of H<sub>2</sub>O are repeated until no further aqueous hydrofluoric acid is formed. The solid is then melted, poured onto in Mg sheet, crushed and stored in a tightly closed can. In addition, SbF<sub>3</sub> can be distilled as described in method I.

Antimony (III) fluoride is kept in glass vessels or iron containers.

PROPERTIES:

Formula weight 178.76. Colorless, deliquescent crystals, readily soluble in water with partial hydrolysis. Solubility in water  $(20^{\circ}C)$  443 g./100 ml.;  $(30^{\circ}C)$  562 g./100 ml.

M.p. 292°C, b.p. 376°C; d. (solid) (20°C) 4.379. Structure: rhombic.

REFERENCES:

- I. O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], 1920, p. 39.
- II. J. Söll. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 276.

# Antimony (V) Fluoride

#### SbF<sub>5</sub>

 $SbF_3 + F_2 = SbF_3$ 178.76 38 216.76

Fluorine is fed into a quartz apparatus (Fig. 126) containing gaseous  $SbF_3$ . The apparatus is heated with a Bunsen burner to bring the  $SbF_3$  to gentle boiling. A fluorine stream of at least 10 g./hour is added through an aluminum tube. The antimony (V) fluoride reacts, at times igniting, and  $SbF_5$  distills. It can then be fractionated in a quartz apparatus.

Antimony (V) fluoride is kept in sealable Al bottles or, if necessary, in quartz vessels. Platinum bottles can also be used.

PROPERTIES:

Colorless, viscous liquid; very reactive. Fizzes when poured into water; is caustic to the skin. Attacks glass, but is only slightly corrosive to Cu and Pb. Inert to quartz, Pt and Al.

M.p. 6°C, b.p. 150°C; d.(liq.)(22°C) 2.993.



Fig. 126. Preparation of antimony (V) fluoride.

REFERENCE:

J. Söll. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 276.

# Antimony Dichloride Trifluoride SbCl<sub>2</sub>F;

 ${}^{}_{178.76} {}^{+}_{70.91} {}^{-}_{249.67} {}^{+}_{178.76} {}^{+}_{70.91} {}^{-}_{249.67} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76} {}^{+}_{178.76}$ 

A weighed amount of  $SbF_3$  is placed in a steel cylinder equipped with a manometer and a needle valve. The container is evacuated, its valve closed, and the container weighed. A  $Cl_2$  cylinder is then connected through a steel capillary, the valve is opened, and  $Cl_{g}$  is allowed to enter the reaction vessel. The  $Cl_{g}$  is quickly absorbed by the SbF<sub>3</sub>, with evolution of heat. From time to time the connection with the  $Cl_{g}$  cylinder is loosened and the reaction vessel is shaken. The  $Cl_{g}$  addition is then resumed. The reaction is terminated as soon as the calculated amount of  $Cl_{g}$  has been absorbed.

Antimony dichloride trifluoride is stored in iron vessels.

Useful as a catalyst for the preparation of numerous organic fluorine compounds.

PROPERTIES:

Viscous liquid.

REFERENCE:

A. L. Henne. Organic Reactions II, p. 61.

## **Bismuth (III) Fluoride**

BiF<sub>3</sub>

 $\begin{array}{r} \text{Bi}(\text{OH})_3 + 3\,\text{HF} = \text{BiF}_3 + 3\,\text{H}_2\text{O} \\ 260.01 & 60.03 & 266.0 & 54.04 \end{array}$ 

Freshly precipitated bismuth hydroxide is evaporated to dryness several times in a Pt dish, using an excess of hydrofluoric acid. It is then calcined in a covered Pt crucible until the HF has completely evaporated. A grayish product remains.

Chemically pure  $BiF_3$  is white. Such high-purity material can be obtained by the reduction of  $BiF_5$  with hydrogen. The H<sub>2</sub> is greatly diluted with  $CO_2$ ; the reaction takes place in a Pt tube at  $80-150^{\circ}C$ .

Use: Preparation of BiF<sub>5</sub>.

PROPERTIES:

Heavy, white (gray if impure) crystalline powder, practically insoluble in water.

M.p. 725-730°C; d. 8.3. Cubic (dimorphous).

REFERENCES:

Muir, Hoffmeister and Robb. J. Chem. Soc. (London) <u>39</u>, 33 (1881). H. v. Wartenberg. Z. anorg. allg. Chem. 244, 344 (1940).

# Bismuth (V) Fluoride BiFs

$$\begin{array}{rrr} \operatorname{BiF}_3 + \operatorname{F}_2 = \operatorname{BiF}_5\\ 266 & 38 & 304 \end{array}$$

A boat made of sintered alumina and containing  $BiF_3$  is pushed with a nickel wire into a sintered alumina tube (see Fig. 127).



Fig. 127. Preparation of bismuth (V) fluoride.

Both ends of the tube are covered with copper caps, which are water cooled and sealed on with picein. The apparatus is best arranged in such a way that it can be rotated approximately  $90^{\circ}$  into a position perpendicular to the axis of the furnace. The  $F_2$  is added through a flexible, 5-m.-long copper capillary. A fluorine stream is passed through the tube at the rate of 20 ml./min., while the oven is heated to about 550°C. At 460°C the BiF<sub>5</sub> starts to sublime from the boat and crystallizes at the end of the reaction tube in thin white needles about 3 mm. long. The sublimation proceeds best at  $500^{\circ}$ C, since at higher temperatures it is so fast that BiF<sub>5</sub> diffuses upstream and crystallizes even at the inlet to the tube.

After the fluorination is finished, the  $F_2$  stream is replaced with a stream of CO<sub>2</sub> or oxygen-free N<sub>2</sub>. The boat is removed from the reaction tube by pulling it with a Ni wire in the direction opposite to the gas flow and placed in the Cu cap. The Cu cap at the other end is replaced by a glass cap (see Fig. 127). The apparatus is now rotated 90°, so that the far end is at the bottom and the gas inlet on top. The clumps of BiF<sub>5</sub> needles in the reaction vessel are scraped off with a Ni wire. They fall through the glass cap into the collecting ampoule, which is then melt-sealed.

The material is best analyzed by reduction of a weighed amount with H<sub>2</sub>. The hydrogen is greatly diluted with CO<sub>2</sub> and the reaction proceeds at  $80-150^{\circ}$ C (1 hour) in a Pt tube. The freshly formed BiF<sub>3</sub> is weighed.

#### PROPERTIES:

White crystals, highly sensitive to moisture. In humid air immediately turns yellow-brown. Reacts with water, sometimes with ignition, forming ozone and  $BiF_3$ . Reacts with kerosene above  $50^{\circ}C$ . Subl. t. appr.  $550^{\circ}C$ .

**REFERENCE:** 

H. v. Wartenberg. Z. anorg. allg. Chem. 224, 344 (1940).

#### Carbon Tetrafluoride

 $\mathbf{CF}_4$ 

I.

$$C + 2F_2 = CF_4$$
  
12 76 88

Degassed activated carbon or carbon black, contained in a nickel boat, is burned in a  $F_2$  stream in the apparatus described for  $SF_6$  (page 169). The reaction must be externally controlled by cooling. The crude  $CF_4$  collects as a liquid in the liquid-oxygen-cooled quartz trap. After the fluorination, while the trap remains cooled with liquid  $O_2$ , the product is removed by suction, using an aspirator. Most of the dissolved gases are thus removed. The product is then passed through a series of fritted gas scrubber bottles containing 20% KOH solution (not NaOH); this extracts  $COF_2$ ,  $SiF_4$  and HF. Finally, the  $CF_4$  is passed over  $P_2O_5$  and recondensed with liquid  $O_2$ . The liquid is carefully fractionated to remove the higher homologs of  $CF_4$  ( $C_2F_6$ ,  $C_3F_8$ ). Then the last traces of dissolved air are removed, using an oil pump, while the trap is cooled with liquid  $O_2$ .

All the apparatus used for the operations following the fluorination can be made of glass.

II.

$$2 \operatorname{CO}_{56} + 4 \operatorname{F}_2 = 2 \operatorname{CF}_4 + \operatorname{O}_2_{56}_{152}_{176}_{176}_{32}$$

This preparation of  $CF_4$  from CO and  $F_2$  has the advantage over method I that the  $CF_4$  obtained is completely free of higher homologs. The preparation is the same as described for  $COF_2$ (page 207). To obtain good yields of  $CF_4$  and as little  $COF_2$  as possible, the CO must be preheated to as high a temperature as possible (appr. 400°C). With a 1000-amp. current in the fluorine cell, the yield is 80-85%, based on CO. With considerably lower currents, for instance, with a current of 10 amp., the yield of  $CF_4$ is no greater than 15%. The crude  $CF_4$  is purified in the same way as described above. The degree of purity of the product can be easily checked by the melting point since this is considerably lowered by dissolved air or  $C_2F_6$ .

Carbon tetrafluoride is stored in glass or steel cylinders. It can be used in vapor-pressure thermometers [W. Menzel and F. Mohry, Z. anorg. allg. Chem. 210, 256 (1933)].

#### PROPERTIES:

Colorless, odorless, thermally very stable gas. Chemically very inert at room temperature.

M.p. -183.6°C, b.p. -127°C; d. (solid)(-195°C) 1.98, d. (liq.) (-183°C) 1.89.

**REFERENCES:** 

- I. O. Ruff and R. Keim. Z. anorg. allg. Chem. <u>192</u>, 249 (1930); 201, 255 (1931).
- II. W. Kwasnik, Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 168; J. Goubeau, W. Bues and W. Kampmann. Z. anorg. allg. Chem. 283, 123 (1956).

#### Trifluoromethane

#### CHF<sub>8</sub>

I.

 $CHI_{s} + 3 HgF = CHF_{s} + 3 HgI$ 393.76 685.83 70.0 982.59

Iodoform, and powdered  $CaF_2$  (as diluent) are ground together in a 20 : 33.4 : 40 ratio and placed in a 100-ml. glass flask, the outlet of which is joined to a liquid nitrogen-cooled trap, which in turn is connected to a drying tube containing  $P_2O_5$ .

The flask is now heated by means of a sulfuric acid bath. The exothermic reaction starts at appr.  $80^{\circ}$ C and the temperature rises to appr.  $180^{\circ}$ C. Crude CHF<sub>3</sub>, colored by iodine, is collected in the trap.

After completion of the reaction, the product is fractionated. The cut coming over at a bath temperature between -40 and  $-30^{\circ}$ C is practically pure CHF<sub>3</sub>. It is washed with 2N NaOH and dried over  $P_2O_5$ . The yield is 45%.

II. 
$$CHCl_3 + 3 HF = CHF_3 + 3 HCl_{119.39} 60 70.0 109.41$$

A stirred stainless steel autoclave, equipped with an iron reflux condenser, is filled with 360 g. of  $CHCl_a$  and 600 g. of  $SbCl_a$  (catalyst). Then 200 g. of anhydrous HF is injected under pressure and the system is heated for 1.5 hours at  $130^{\circ}$ C. The pressure increases to 75 atm. gage. The pressure is now gradually released through a valve above the condenser. The vented gases are passed through ice water and dilute NaOH, dried over P<sub>2</sub>O<sub>5</sub>, and fractionated.

The experiment can be repeated with the same catalyst if each charge subsequently added to the autoclave consists of 360 g. of CHCl<sub>3</sub> and 60 g. of HF. The yield is 95%.

This fluorination method, in which the catalyst is  $SbF_3Cl_2 \cdot 2HF$ , formed as an intermediate, is widely applicable. It can also be used for the preparation of  $CClF_3$ ,  $CCl_2F_2$ ,  $C_2Cl_3F_3$ ,  $C_2Cl_2F_4$  and  $C_2HCl_3F_2$ .

Trifluoromethane can be stored in a glass flask or a gasometer over water.

SYNONYM:

Fluoroform

PROPERTIES:

Colorless gas, thermally stable up to 1150°C. Chemically unusually stable.

M.p. --160°C, b.p. --84.4°C; d. (liq.) (--100°C) 1.52, d. (solid) 1.935.

**REFERENCES:** 

I. O. Ruff. Ber. dtsch. chem. Ges. 69, 299 (1936).

II. B. Whallay. J. Soc. Chem. Ind. 66, 429 (1947).

## Trifluoroiodomethane

CIF<sub>3</sub>

I.

 $5 \operatorname{CI}_4 + 3 \operatorname{IF}_5 = 5 \operatorname{CIF}_3 + 9 \operatorname{I}_2 \\ 5 \cdot 519.68 \quad 3 \cdot 221.92 \quad 5 \cdot 195.92 \quad 18 \cdot 126.92$ 

A glass flask provided with a gas outlet is filled with 80 g. (0.153 mole) of  $CI_4$  and 30 g. (0.135 mole) of  $IF_5$ . The gas outlet is connected via short rubber tubes to several gas traps cooled with liquid nitrogen. Agitation of the vessel produces vigorous evolution of gas. When the reaction subsides, the system is heated for 30 min. at 90-100°C. The condensate in the gas traps is then washed with 5% NaOH and fractionated. The yield is 90%.

II.  $CF_3COOAg + I_2 = CF_3I + AgI + CO_2$ 220.89 253.84 195.92 234.80 44.01

The starting material, silver trifluoroacetate, is first produced by adding  $Ag_2O$  to 50% trifluoroacetic acid solution and evaporating the mixture to dryness under vacuum.

The powdered silver trifluoroacetate (100 g.) is mixed with 110-300 g. of powdered iodine and poured into a glass tube closed at one end. The tube is placed horizontally and the open end connected to a trap cooled with ice water; this in turn is connected to two Dry-Ice-cooled traps and a water-filled bubble counter. The mixture is then gradually heated with a gas flame to above  $100^{\circ}C$ ; the rate of heating should be controlled by observing the flow through the bubble counter. Iodine collects in the first trap; CIF<sub>3</sub> in the last. The latter is washed with dilute NaOH and purified by fractionation. The yield is 80-95%.

Trifluoroiodomethane is stored in glass ampoules.

#### PROPERTIES:

Colorless, light-sensitive gas. Evolves  $CF_3 \cdot \text{radicals}$  when heated or irradiated with UV light and is therefore useful in the synthesis of numerous compounds of the type  $CF_3(CF_2)_n \cdot X$ , as well as organometallic and organometalloid compounds.

#### REFERENCES:

- I. A. A. Banks, H. J. Emeléus, R. N. Haszeldine and V. Kerrigan. J. Chem. Soc. (London) <u>1948</u>, 2188.
- II. R. N. Haszeldine. J. Chem. Soc. (London) <u>1951</u>, 584; A. L. Henne and W. G. Finnegan. J. Am. Chem. Soc. <u>72</u>, 3806 (1950).

## Carbonyl Fluoride

COF<sub>2</sub>

I.

$$\begin{array}{c} \operatorname{CO}_{28} + \operatorname{F}_{2} = \operatorname{COF}_{2}_{28} \\ 38 & 66 \end{array}$$

A copper cylinder, equipped at the bottom with a detachable burner, is used as the reaction vessel. Two observation tubes, each consisting of a 30-cm.-long copper tube with a quartz window (rubber gasket seal), allow observation of the flame. The vessel is wrapped with towels or muslin bandages to permit thorough wetting of the apparatus wall by the cooling water running over it.

The reaction products pass through a short condenser for initial cooling. This in turn is connected to two quartz traps cooled with liquid  $O_2$  (Fig. 128).

The input CO is purified by washing with pyrogallol solution and concentrated  $H_2SO_4$ ; it is then passed through a  $P_2O_5$  drying



Fig. 128. Preparation of carbonyl fluoride.

tube and a flowmeter. Finally a safety relief vessel to accomodate excess pressure is connected to help recognize immediately any plugging of the apparatus.

The  $F_2$  is taken directly from an electrolysis cell and passed through an iron condenser coil, cooled to  $-78^{\circ}C$  with Dry Ice in order to separate any HF.

To start the operation,  $F_2$  is passed through the entire apparatus until a gas flame or, better, an oil-soaked piece of fabric can be ignited by the exit gases. Then the burner is quickly unscrewed and the CO stream ignited. The burner is replaced in the reactor with its flame adjusted to a small size so that the CO continues to burn in the fluorine stream. The CO and  $F_2$  streams are then controlled to give a constant excess of fluorine. A CO flow of 2.5 liters/hour per 10 amp. of current in the fluorine cell is optimum.

If the opposite procedure is used and the  $F_2$  stream is burned in a CO atmosphere, there is a risk of a violent explosion should the flame go out unexpectedly.

Carbon tetrafluoride can be prepared in the same apparatus (see page 203). However, to obtain high yields of  $CF_4$ , a high flow rate of the input gases is necessary (for instance, 1000 amp. current in the  $F_2$  cell and 250 liters of CO/hour) and the CO must be preheated to appr. 400°C. This is best accomplished by heating the CO inlet tube with a Bunsen burner.

At the temperature of liquid  $O_2$ , the product condensed in the first quartz trap is part solid and part liquid  $(CF_4)$ . To isolate the  $COF_2$ , the product should be distilled into a small steel cylinder after stripping from it the major part of the dissolved gases ( $F_2$ , air) at  $-183^{\circ}C$  for 1/2 hour with a water aspirator. Following the distillation the cylinder is turned upside down and the liquid portion  $(CF_4)$  pumped out with an oil pump. The solid  $COF_2$  remains in the steel container. This procedure is accomplished in appr. two minutes and yields 85% pure  $COF_2$ . If a low-temperature filter is

available, the separation of  $COF_2$  and  $CF_4$  can also be accomplished at -183°C. Finally, the material is fractionated in a quartz apparatus. A 97% pure product is obtained.

Carbonyl fluoride is stored in steel cylinders.

II.  $COF_2$  is also conveniently prepared in a completely  $CF_4$ -free form via the reaction of  $BrF_3$  with CO. This procedure is described in detail under the preparation of carbonyl bromofluoride (p. 210).

To isolate the  $COF_2$ , the reaction product, which is colored yellow by the bromine, is passed over Sb powder and recondensed. The mixture is then fractionated at atmospheric pressure, and pure  $COF_2$  comes over between -85 and -60°C. This procedure is very convenient since it can be left virtually unattended.

SYNONYM:

Fluoroformyl fluoride, carbonyl difluoride.

PROPERTIES:

Colorless gas, very hygroscopic, pungent odor. Instantly hydrolyzed by water.

M.p.  $-114.0^{\circ}$ C, b.p.  $-83.1^{\circ}$ C; d(solid) ( $-190^{\circ}$ C) 1.388, d(liq.) ( $-114^{\circ}$ C) 1.139.

REFERENCES:

- I. O. Ruff and G. Miltschitsky. Z. anorg. allg. Chem. <u>221</u>, 154 (1935); W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 168.
- II. W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 242.

# Carbonyl Chlorofluoride

#### COCIF

 $\begin{array}{c} \text{ClF} + \text{CO} = \text{COClF} \\ 54.46 \quad 28 \quad 82.46 \end{array}$ 

Streams of ClF and CO are mixed at  $-18^{\circ}$ C in an iron reaction vessel (see Fig. 129). The CO must always be present in excess. The slower the rate of reaction the greater the yield of COClF. The reaction gases are condensed in a quartz trap at  $-196^{\circ}$ C. The second quartz trap is used to exclude atmospheric moisture.

After the reaction is finished, the yellow product is repeatedly passed over Sb powder and finally distilled over the Sb. The white



Fig. 129. Preparation of carbonyl chlorofluoride.

product is then fractionated at atmospheric pressure. The first cut is  $COF_2$ , the last  $COCI_2$ . The middle fraction (-50° to -30°C) is COCIF. It can be made extremely pure by repeated fractionation. The yield is 85-90%, based on CIF.

Carbonyl chlorofluoride is preferably stored in quartz ampoules cooled in liquid nitrogen. It can also be stored under pressure in cylinders made of type 316 stainless steel.

SYNONYM:

Chlorofluorophosgene.

PROPERTIES:

Colorless. Odor almost indistinguishable from phosgene.

M.p.  $-148^{\circ}$ C, b.p.  $-47.2^{\circ}$ C; d (liq.) ( $-78^{\circ}$ C) 1.506, ( $0^{\circ}$ C) 1.323, (18°C) 1.277. V. p. (19°C) 12 atm. gage; t<sub>cr</sub> +85°C.

Stable at room temperature both as a gas and a liquid. Hydrolyzed by water within half an hour. Absorbed immediately by NaOH solution, evolving heat and leaving no residue. Glass is stable to it for weeks but becomes covered with a cloudy film. Quartz is more stable but is is also slowly covered with a cloudy film. Attacks Hg. After exposure for a week, rubber becomes somewhat hard. Stainless steels 304 and 316, brass and aluminum are inert to COCIF; Ni, Monel, Sn, Zn and electron (Mg-Al) alloys have moderate resistance; Fe, Cu, Pb and Ag show little resistance.

**REFERENCE:** 

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 242.

## **Carbonyl Bromofluoride**

#### COBrF

$$BrF_{3} + 2 CO = COBrF + COF_{2}$$
  
136.9 56.0 126.9 66

An iron wash bottle (Fig. 130) with a screw cap is filled with BrF<sub>3</sub>, the cover is screwed on, and the vessel is cooled with ice water. Two quartz traps cooled with liquid N<sub>2</sub> are then connected to it. The first trap is the usual condensation vessel; the second is used solely to exclude atmospheric moisture. The CO, which has been purified by passage through pyrogallol solution, concentrated H<sub>2</sub>SO<sub>4</sub> and  $P_3O_5$ , is bubbled through the BrF<sub>3</sub>. The system evolves heat. The CO flow is so regulated that the temperature of the BrF, is kept between +8 and +30°C. The BrF<sub>3</sub> freezes below 8°C and the reaction proceeds explosively at too high a temperature. The product condensing in the first trap is yellowish. After the reaction is complete the product is passed over Sb powder to remove the Br and then fractionated in a quartz apparatus. Pure  $COF_2$  comes over in the first fraction (see p. 208) between -85 and -60°C, and COBrF is collected from -30 to  $-15^{\circ}$ C. The latter can be purified by refractionation. The yield is greater than 90%. based on  $Br F_3$ .



Fig. 130. Preparation of carbonyl bromofluoride.

Carbonyl bromofluoride is preferably stored in quartz ampoules kept in liquid nitrogen. It can also be kept at room temperature in quartz containers or type 304 stainless steel cylinders, but it becomes yellow-brown with time and must be redistilled before use.

SYNONYMS:

Bromofluorophosgene.

PROPERTIES:

Colorless gas.

M.p. --120°C, b.p. --20.6°C; v.p. (18°C) 3.65 atm. gage;  $t_{\rm Cr}$  +124°C,  $p_{\rm Cr}$  appr. 61 atm.; d (liq.) (0°C) 1.944.

Gaseous COBrF is thermally stable up to  $125^{\circ}$ C. Liquid COBrF decomposes slowly at room temperature. Water causes quantitative hydrolysis to CO<sub>2</sub> + HBr + HF in appr. 30 minutes. Instantly absorbed by NaOH solution. Glass is stable to it for some time. Attack by liquid COBrF causes rubber to become black and brittle. Attacks Fe and Hg.

Odor similar to that of phosgene, but with some experience it can readily be differentiated.

REFERENCE:

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 242.

## Carbonyl Iodofluoride

#### COIF

 $\begin{array}{c} \mathbf{IF}_{5} + 3\,\mathbf{CO} = \mathbf{COIF} + 2\,\mathbf{COF}_{2} \\ 221.93 & 84.0 & 173.93 & 132.0 \end{array}$ 

A one-liter rocker bomb is charged with 50 g. of  $IF_5$ ; it is pressurized to 120 atm. with CO and rocked in an inclined position for a week. The pressure is then relieved until atmospheric pressure is reached, thus removing the COF<sub>2</sub> and the excess CO. Next, a quartz trap containing some Sb powder and cooled with liquid N<sub>2</sub> is connected at the valve of the autoclave. The system is evacuated for an hour to about 200 mm. The COIF thus distills over and is condensed in the trap. The autoclave can then be refilled with CO without recharging the  $IF_5$ .

The collected condensate is distilled from the Sb in the trap. The COF<sub>2</sub> is removed below  $-15^{\circ}$ C. The distillation is then continued at reduced pressure (appr. 300 mm.) because of the instability of the COIF. It comes over between -15 and  $+20^{\circ}$ C. It is redistilled at reduced pressure over Sb. The main products of this reaction are  $I_2$  and COF<sub>2</sub>. The yield is 12%, based on IF<sub>5</sub>.

Carbonyl iodofluoride can be stored only in quartz ampoules under Dry Ice or, better, liquid nitrogen.

SYNONYM:

Iodofluorophosgene.

PROPERTIES:

Colorless if pure. Choking odor, similar to COBrF, quite distinct from COCl<sub>2</sub>.

M.p.  $-120^{\circ}$ C, b.p.  $-20.6^{\circ}$ C; v.p. (18°C) 3.65 atm. gage; tor 124°C. pcr appr. 61 atm.; d (liq.) (0°C) 1.944.

Above -20°C, COIF decomposes perceptibly with liberation of iodine. Gaseous COIF also decomposes at room temperature. Slowly hydrolyzed by water, similarly to COBrF. Absorbed completely by NaOH. Quartz and glass become coated with a yellow substance on contact with the liquid at room temperature.

REFERENCE:

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 242.

#### Silicon Tetrafluoride

SiF₄

I.

 $2 \operatorname{CaF}_2 + 2 \operatorname{H}_2 \operatorname{SO}_4 = 2 \operatorname{CaSO}_4 + 4 \operatorname{HF}$ 272.25 156.14 196.15 80.04

> $4 \mathrm{\,HF} + \mathrm{SiO}_2 = \mathrm{SiF}_4 + 2 \mathrm{\,H}_2\mathrm{O}$ 60.05 104.06 80.04 36.03

Powdered calcium fluoride is fumed with HF in a Pt dish in order to remove carbonates.

A stoichiometric mixture of calcium fluoride powder, an excess of quartz sand of highest purity (99.9%), and concentrated H<sub>2</sub>SO<sub>4</sub> are placed in the reaction flask of an all-glass apparatus (see Fig. 131), and gently warmed on a sand bath. The evaporating SiF<sub>4</sub> passes through a vertical water-cooled condenser and a trap cooled with Dry Ice-acetone mixture to remove possible impurities (HF), and is finally condensed in a trap cooled with liquid nitrogen. In order to exclude moist air, a P<sub>2</sub>O<sub>5</sub> drying tube is connected to the system. The product can be purified by sublimation in a closed glass vessel or distillation under slight pressure, in which case the first and last cuts can be discarded.

II.  $H_2SiF_6$  (+ conc.  $H_2SO_4$ ) = SiF<sub>4</sub> + 2 HF 144.06 104.06

An iron vessel (Fig. 132) is substituted for the glass reaction flask of method I. The wrought-iron container holds one liter of

40



Fig. 131. Preparation of silicon tetrafluoride (I).

Fig. 132. Preparation of silicon tetrafluoride (II).

60% H<sub>2</sub>SiF<sub>6</sub>. Two liters of concentrated H<sub>2</sub>SO<sub>4</sub> are added dropwise through a dropping funnel inserted into the container via a rubber stopper. The iron extension tube of the funnel extends into an iron tube which is closed at the bottom and from the top of which the H<sub>2</sub>SO<sub>4</sub> overflows. The HF formed during the reaction is completely retained by the concentrated H<sub>2</sub>SO<sub>4</sub>.

Silicon tetrafluoride can be stored in a glass flask with a stopcock, in gasometers over Hg or concentrated  $H_2SO_4$ , or in steel cylinders.

SYNONYM:

Tetrafluorosilane.

PROPERTIES:

Colorless gas, very hygroscopic, forms a dense fog in humid air, is rapidly cleaved by water, does not attack stopcock grease.

Subl. t.  $-95^{\circ}$ C, m.p. (under pressure)  $-90.2^{\circ}$ C; d. (liq.) ( $-88^{\circ}$ C) 1.590; t<sub>cr</sub>  $-1.5^{\circ}$ C; p<sub>cr</sub> 50 atm. gage.

**REFERENCES:** 

- I. L. Lebouché, W. Fischer and W. Biltz. Z. anorg. allg. Chem. 207, 64 (1932); O. Ruff and E. Ascher. Z. anorg. allg. Chem. <u>196</u>, 413 (1931).
- II. J. Söll. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 257.

## Trifluorosilane

#### SiHF<sub>3</sub>

 $\begin{array}{rrr} 4 \, \text{SiHCl}_3 \, + \, 3 \, \text{TiF}_4 \, = \, 4 \, \text{SiHF}_3 \, + \, 3 \, \text{TiCl}_4 \\ 541.76 & 371.7 & 344.28 & 569.19 \end{array}$ 

Trichlorosilane and TiF<sub>4</sub> are heated for 18 hours in an autoclave on an oil bath at 100-200°C. If necessary, the reaction can be carried out in a sealed pressure tube. After cooling, the autoclave is slowly vented and the exit gases are collected in a quartz or glass trap cooled in liquid N<sub>2</sub>. The mixture is then fractionated. The residue in the autoclave consists of TiF<sub>4</sub> and TiCl<sub>4</sub>. Since pure SiHF<sub>3</sub> decomposes slowly even at room temperature it should be kept in Dry Ice or liquid N<sub>2</sub>.

SYNONYM:

Silicofluoroform.

PROPERTIES:

Formula weight 86.07. Colorless, flammable gas; forms an explosive mixture with air. Decomposes slowly at room temperature; decomposes rapidly to  $H_2$ , Si and SiF<sub>4</sub> if heated to 400°C. Hydrolyzed by water. Decomposes alcohol and ether; reduces concentrated nitric acid.

M.p.  $-110^{\circ}$ C, b.p.  $-80^{\circ}$ C.

REFERENCE:

O. Ruff and C. Albert. Ber. dtsch. chem. Ges. 38, 56 (1905).

## Hexafluorosilicic Acid

#### H<sub>2</sub>SiF<sub>6</sub>

 $\begin{array}{rrrr} 6\,\mathrm{HF}\,+\,\mathrm{SiO_2}\,=\,\mathrm{H_2SiF_6}\,+\,2\,\mathrm{H_2O}\\ 120 & 60.06 & 144.03 & 36.03 \end{array}$ 

I. Small portions of quartz powder (99.9%) are added to 70-95% hydrofluoric acid, containing a small amount of  $H_2SiF_8$ . The addition is carried out in an iron vessel and proceeds until no further dissolution of the quartz occurs. The reaction must be moderated by cooling with ice. The addition of  $H_2SiF_8$  is necessary for a smooth initiation of the reaction. After the reaction is terminated, the

excess of quartz powder is left to settle and the 60-70% H<sub>2</sub>SiF<sub>g</sub> is decanted. The material is best stored in iron containers. Concentrated hexafluorosilicic acid solidifies at appr.  $19^{\circ}$ C; the tetrahydrate crystallizes out and must be melted by gentle warming before the container can be emptied.

Other preparative methods: II. Addition of  $SiF_4$  to water. III. Reaction of concentrated  $H_2SO_4$  with  $BaSiF_6$ .

Use: Preparation of fluorosilicates and SiF<sub>4</sub>.

SYNONYMS:

Fluosilicic acid, fluorosilicic acid, silicofluoric acid.

PROPERTIES:

Colorless liquid. Anhydrous  $H_2SiF_6$  is 50% dissociated to  $SiF_4$  and HF even at room temperature. Can be distilled without decomposition only as a 13.3% aqueous solution. Aqueous  $H_2SiF_6$  does not attack glass.

Specific gravity of aqueous solutions at  $17.5^{\circ}C.6\%$ , 1.049; 20%, 1.173; 34%, 1.314.

REFERENCES:

- I. J. Söll. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 257.
- II. W. Hempel. Ber. dtsch. chem. Ges. 18, 1438 (1885).
- III. E. Baur and A. Glaessner. Ber. dtsch. chem. Ges. <u>36</u>, 4215 (1903).

## Germanium Tetrafluoride

GeF<sub>4</sub>

$$\begin{array}{r} BaGeF_{6} = GeF_{4} + BaF_{2} \\ 323.96 & 148.60 & 175.36 \end{array}$$

The complex salt BaGeF<sub>g</sub> is precipitated in a Pt dish by adding BaCl<sub>2</sub> to a solution of GeO<sub>2</sub> in hydrofluoric acid. The granular precipitate is washed, dried, placed in a quartz tube and heated in a N<sub>2</sub> stream. Formation of GeF<sub>4</sub> starts at appr. 500°C and proceeds vigorously at 700°C (apparatus for SF<sub>6</sub>, p.169). The temperature is slowly increased to 1000°C. The exit gases are passed through a quartz trap cooled with liquid N<sub>2</sub> so that the GeF<sub>4</sub> condenses and solidifies. The product is then fractionated in a quartz apparatus; the first cut is SiF<sub>4</sub>. The yield is 87%.

Germanium tetrafluoride is stored in glass bottles or, better, sealed under pressure in quartz ampoules.

PROPERTIES:

Colorless gas, thermally stable to  $1000^{\circ}$ C; fumes strongly in air; has a pungent garlic odor; attacks the respiratory organs and causes hoarseness. Hydrolyzed in water to GeO<sub>2</sub> and H<sub>2</sub>GeF<sub>6</sub>. Attacks Hg, but not glass, if absolutely anhydrous. Attacks stop-cock grease.

M.p.  $-15^{\circ}$ C, subl. t.  $-36.5^{\circ}$ C; d(liq.) (0°C) 2.162, d(solid) (-195°C) 3.148.

**REFERENCES:** 

- L. M. Dennis and A. W. Laubengayer. Z. phys. Chem. <u>130</u>, 520 (1927).
- L. M. Dennis. Z. anorg. allg. Chem. 174, 119 (1928).
- L. Le Boucher, W. Fischer and W. Biltz. Z. anorg. allg. Chem. 207, 65 (1932).

#### Potassium Hexafluorogermanate

#### K<sub>2</sub>GeF<sub>6</sub>

 $\begin{array}{c} \text{GeO}_2 + 6\,\text{HF} + 2\,\text{KCl} = \text{K}_2\text{GeF}_6 + 2\,\text{HCl} + 2\,\text{H}_2\text{O} \\ 104.6 & 120.0 & 149.1 & 264.8 & 72.9 & 36.0 \end{array}$ 

Two parts of GeO<sub>2</sub> are dissolved in 12 parts of 20% HF in a Pt dish and 3 parts of a concentrated KCl solution are added. The liquid solidifies to a gel which on stirring again becomes liquid and precipitates as a dense crystalline powder. The solid is filtered, washed consecutively with small amounts of water and alcohol, and dried. A solution of  $K_2CO_3$  can be employed instead of KCl.

PROPERTIES:

White crystalline powder, nonhygroscopic. Recrystallization from water yields plates.

M.p.  $\sim 730^{\circ}$ C; b.p.  $\sim 835^{\circ}$ C; Solubility: 1 g. in 184.6 g. H<sub>2</sub>O(18°C), in 34.0 g. H<sub>2</sub>O(100°C). Crystalline form: hexagonal.

REFERENCES:

C. Winkler, J. prakt. Chem. [2] <u>36</u>, 199 (1887).
J. H. Müller. J. Amer. Chem. Soc. <u>43</u>, 1089 (1921).
G. Krüss and O. Nilson. Ber. dtsch. chem. Ges. <u>20</u>, 1697 (1887).

# Tin (II) Fluoride SnF.

 $\begin{array}{c} \text{SnO} + 2\,\text{HF} = \text{SnF}_2 + \text{H}_2\text{O} \\ \text{i}43.70 \quad 40 \quad 156.70 \quad 18.01 \end{array}$ 

Tin (II) oxide is dissolved in 40% HF in a Pt dish and evaporated to dryness with exclusion of air.

Better defined crystals are obtained if 67.4 g. of SnO (0.5 mole) is dissolved in 15-20 ml. of degassed water in a 200-ml. polyethylene beaker. The contents are heated on a steam bath to  $60^{\circ}$ C in an O<sub>2</sub>-free nitrogen atmosphere and 46 g. of 48% hydrofluoric acid (1.1 moles) is added slowly and dropwise while the beaker is rotated. The reaction evolves heat. When all the solid is dissolved, the beaker is placed in a desiccator and cooled, so that crystals separate. After two hours, the mother liquor is decanted into a second beaker. Both beakers are then placed in a desiccator over a mixture of CaCl<sub>2</sub> and KOH (1 : 1). After two days, this drying agent is removed and Mg (ClO<sub>4</sub>)<sub>2</sub> is substituted. After an additional four days, the mother liquor is again decanted and a second crop of crystals thus obtained. It is dried in the same manner as the first. The yield is 86%.

PROPERTIES:

Colorless prisms, soluble in water, yielding a clear solution. Crystal structure: monoclinic.

M.p. 210-215°C.

**REFERENCES:** 

- J. L. Gay-Lussac and L. J. Thénard. Mém. phys. Chim. 2, 317 (1809).
- H. Nebergall, J. C. Muhler and H. G. Day. J. Amer. Chem. Soc. <u>74</u>, 1604 (1952).

#### Tin (IV) Fluoride

SnF₄

 $SnCl_4 + 4 HF = SnF_4 + 4 HCl_{260.53} 80 194.7 145.84$ 

In the same way as described for  $TiF_4$  (page 250),  $SnCl_4$  is added dropwise to double the theoretical amount of anhydrous HF,

thus forming the complex  $\mathrm{SnCl}_4 \cdot \mathrm{SnF}_4$ . A copper reflux condenser (ice-salt mixture is used as cooling agent) is attached to the reaction vessel and the system is heated in the presence of excess HF until HCl ceases to evolve. The HF is then distilled off through an inclined condenser. The temperature is finally raised to 130-220°C so that the complex  $\mathrm{SnCl}_4 \cdot \mathrm{SnF}_4$  is cleaved and  $\mathrm{SnCl}_4$  distills over. A distillation head is then placed on the reaction vessel and  $\mathrm{SnF}_4$  is sublimed at red heat. The inclined section of the head should preferably be covered with wet asbestos paper. The  $\mathrm{SnF}_4$  is immediately charged into closed Fe or Cu containers.

PROPERTIES:

Snow-white, starlike crystal clumps; extremely hygroscopic; dissolves in water with vigorous fizzing. Subl. t. 705°C:d(19°C) 4.78.

REFERENCE:

O. Ruff and W, Plato. Ber. dtsch. chem. Ges. 37, 673 (1904).

#### Lead (II) Fluoride

#### PbF<sub>2</sub>

 $\begin{array}{l} PbCO_3 + 2\,HF = PbF_2 + H_2O + CO_2 \\ 267.21 & 40 & 245.21 & 18.01 & 44.00 \end{array}$ 

Small portions of nitrate- and acetate-free  $PbCO_3$  are added to hydrofluoric acid contained in a Pb or Pt dish. The HF must be present in excess. The mixture is heated for about one day until  $CO_2$  ceases to evolve. The excess of acid is then decanted and the residue evaporated to dryness on a hotplate. The product is then rapidly melted by placing the Pt dish for a few minutes in an electric furnace which is preheated to red heat. The lead hydrofluoride is thereby decomposed.

Because of its impurities, this PbO is less suitable as a starting material for PbF<sub>2</sub> than  $Pb(OH)_2$ .

PROPERTIES:

White crystalline powder. Dimorphous; rhombic  $\alpha$ -PbF<sub>2</sub> (lead chloride type) changes above 316°C into cubic  $\beta$ -PbF<sub>2</sub> (fluorite type).

M.p. 824 °C, b.p. 129 °C; d 824. Solubility in water (0 °C), 0.057 g./100 g. H<sub>2</sub>O; (20 °C), 0.065 g./100 g. H<sub>2</sub>O. The presence of HNOs or nitrates increases the solubility.

REFERENCE:

O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Springer, Berlin, 1920, p. 33.

## Lead (IV) Fluoride

#### **PbF**₄

 $PbF_2 + F_2 = PbF_4$ 245.21 38 283.21

The apparatus described under BiF<sub>5</sub> (p. 202) contains an alundum boat used for fluorination of PbF<sub>2</sub> at 300°C. The F<sub>2</sub> is initially diluted with CO<sub>2</sub> or N<sub>2</sub>, but its concentration in the gas mixture is slowly increased while the temperature is gradually raised to 500°C. The major portion of the PbF<sub>4</sub> remains in the boat in the form of 1-2 mm. -long needles.

After the fluorination is terminated, the boat with the  $PbF_4$  is pulled into the glass cap placed on the reaction vessel, and the product is scraped out with a Ni wire. The solid drops into the glass ampoules, which are immediately sealed off.

PROPERTIES:

White crystalline substance, very sensitive to moisture, immediately discolors in air yielding brown PbO<sub>2</sub>.

M.p. 600°C; d 6.7; tetragonal crystals.

**REFERENCE:** 

H. v. Wartenberg. Z. anorg. allg. Chem. 244, 339 (1940).

#### Boron Trifluoride

#### BF<sub>3</sub>

I.

A mixture of 80 g. of dried or, preferably, melted  $\text{KBF}_4$  and 30 g. of  $B_2O_3$  is heated to about 600°C in an inclined iron tube (40 cm. long, 3 cm. diameter), which is sealed at one end. The other end of the iron tube is closed by a flange sealed with a copper gasket. An appr. 10-mm.-diameter iron tube is welded into an opening in the flange and is connected to a drying tube filled with glass wool, which acts as a dust filter. The drying tube is in turn joined to a quartz or glass trap cooled in liquid nitrogen. The apparatus ends in a drying tube filled with freshly dried KF. The yield is 17 g. of  $BF_3$ . This can be purified by repeated fractional distillation.

II. 
$$6 \text{ NaBF}_4 + \text{B}_2\text{O}_3 + 6 \text{H}_2\text{SO}_4 = 8 \text{ BF}_3 + 6 \text{ NaHSO}_4 + 3 \text{H}_2\text{O}_{658.92} 69.64 588.40 542.56 720.36 54.04$$

A mixture of 300 g. of NaBF<sub>4</sub>, 50 g. of  $B_2O_3$  and 300 ml. of concentrated  $H_2SO_4$  is carefully heated in a one-liter flask provided with a ground-glass joint (see Fig. 133) until gas evolution starts. Only then can more heat be applied. The exit gas passes through a condenser, then through an absorption tube filled with  $B_2O_3$ which has been interspaced with glass wool, and finally it is condensed in a trap at -196°C. A KF drying tube is placed at the end of the system in order to exclude moisture.



trifluoride.

The advantage of this method of preparation is that the residues are water soluble and the reaction vessel can be easily cleaned. According to Ryss and Polyakova, the best  $BF_3$  yield (80%) is obtained at  $180^{\circ}C$  with 105.9% sulfuric acid (oleum) in a 200%

III.  $\begin{array}{ccc} H_{3}BO_{3} + 3 HSO_{3}F = BF_{3} + 3 H_{2}SO_{4} \\ 61.84 & 300.22 & 67.82 & 294.24 \end{array}$ 

Concentrated  $H_2SO_4$  is placed in an iron reaction vessel, which has one gas and two addition nozzles on top and one outlet nozzle

220

excess.

(with a valve) at the bottom. A solution of 20-25% boric acid in concentrated  $H_2SO_4$  and  $HSO_3F$  is added at 85 to  $135^{\circ}C$ . The  $BF_3$  is slowly liberated. The  $H_2SO_4$  which accumulates may be removed from time to time at the bottom and may be used to dissolve the boric acid.

Further preparative methods: IV. Thermal decomposition of diazonium fluoroborates [G. Balz and G. Schiemann, Ber. dtsch. chem Ges. 60, 1186 (1927)].

V. A mixture of 40 g. of KBF<sub>4</sub>, 8 g. of  $B_2O_3$  and 120 ml. of concentrated sulfuric acid is heated to  $270^{\circ}C$  on a sand bath in a 300-ml. flask equipped with ground-glass joints [P. Baumgarten and H. Henning. Ber. dtsch. chem. Ges. 72, 1747 (1931)].

The older method for preparing  $BF_3$  starting with  $CaF_2$  is not recommended, since the yields are low and the product is contaminated with  $SiF_4$ .

The product is stored in glass containers over Hg or in steel cylinders.

Derivatives:  $BF_3 \cdot (OC_2H_5)_2 p. 786$  $BF_3 \cdot NH_3 p. 785$  $BF_3 \cdot 2H_2 O p. 784$  $H[BF_2(OH)_2] p. 784$  $n-C_4H_9BF_2 p. 802$ 

PROPERTIES:

Colorless, asphyxiating gas, fumes in moist air, thermally very stable.

M.p.  $-128^{\circ}$ C, b.p.  $-101^{\circ}$ C; t<sub>cr</sub>  $-12.25^{\circ}$ C; p<sub>cr</sub> 50.2 atm. gage; d (liq.) ( $-128^{\circ}$ C) 1.769; d (solid) ( $-150^{\circ}$ C) 1.87.

Hydrolyzes in water to give  $H_3BO_3$  and  $HBF_4$ . The gas attacks rubber. Rubber tubing and stoppers should therefore be avoided in apparatus used in its preparation.

REFERENCES:

I. W. Hellriegel. Ber. dtsch. chem. Ges. 70, 689 (1937).

- II. H. S. Booth and K. S. Willson. J. Amer. Chem. Soc. 57, 2273 (1935); I.G. Ryss and Y.M. Polyakova. Zh. Obshch. Khim. 19 (81), 1596 (1949) (Chem. Zentr. 50. II. 1329).
- III. U. S. Patent 2, 416, 133.

## Fluoroboric Acid

#### HBF<sub>4</sub>

 $\begin{array}{r} H_{3}BO_{3} + 4 HF = HBF_{4} + 3 H_{2}O \\ 61.82 & 80.04 & 87.82 & 54.04 \end{array}$ 

A slightly larger than stoichiometric quantity of  $H_3BO_3$  is added in small portions to an ice-cooled iron reaction vessel containing 70-90% hydrofluoric acid. The reaction is highly exothermic. After the reaction is completed, the excess  $H_3BO_3$  is allowed to settle out and the pure HBF<sub>4</sub> is decanted.

Fluoroboric acid is stored in glass containers.

PROPERTIES:

Colorless liquid, does not attack glass at room temperature. Decomposes on heating with water, forming oxyfluoroboric acids. Toxic and inhibits fermentation even when present in traces.

**REFERENCES:** 

Mathers, Stewart, Housemann and Lee. J. Amer. Chem. Soc. 37, 1516 (1915).

F. Fichter and K. Thiele. Z. anorg. allg. Chem. 67, 302 (1910).

# Sodium Fluoroborate NaBF4

$2 H_{3}BO_{3}$	+8 HF	$+ Na_2CO_3$	$= 2 \text{NaBF}_4$	+7 H <sub>2</sub> O	$+ CO_2$
123.64	160.08	105.99	219.63	126,1	44.0

Boric acid (6.2 g.) is added, with cooling, to 25 g. of 40% hydrofluoric acid contained in a Pt dish. The mixture is left standing for six hours at room temperature, then cooled with ice, and 5.3 g of dry  $Na_2CO_3$  is added. The solution is then evaporated until crystallization starts. The salt can be recrystallized from water, whereby large, beautiful single crystals can be obtained. The  $NaBF_4$  is finally dried under vacuum.

SYNONYMS:

Sodium fluoborate, sodium borofluoride.

**PROPERTIES:** 

Formula weight 109.815. Colorless salt; crystallizes in the anhydrous form as clear, orthogonal, stubby prisms. Anhydrous NaBF<sub>4</sub> does not etch glass. Readily soluble in water. Rhombic crystals, isodimorphous with NaClO<sub>4</sub>.

**REFERENCE:** 

G. Balz and E. Wilke-Dörfurt, Z. anorg. allg. Chem. 159, 197 (1927).

222

# Potassium Fluoroborate KBF4

# $H_3BO_3 + 4 HF + KOH = KBF_4 + 4 H_2O_{61.82} 80.04 56.11 125.92 72.05$

Boric acid (6.2 g.) is added to 25 g. of 40% hydrofluoric acid solution contained in an ice-cooled platinum dish. The solution is allowed to stand at room temperature for six hours. At the end of this period it is again chilled with ice, and 5N KOH solution is added with constant stirring until the color of methyl orange changes. Crystalline  $KBF_4$  precipitates out at the same time. The mother liquor and subsequent water washings are decanted and the crystals dried under vacuum. The yield is 90%.

#### PROPERTIES:

White, crystalline salt, nonhygroscopic.

M.p. 530°C,  $d_4^{2\circ}$  2.505. Solubility in water (20°C) 0.45; (100°C) 6.3 g./100 ml. Dimorphous: rhombic-bipyramidal and cubic structures (trans. temp. 276-280°C).

**REFERENCES:** 

D. Vorländer, J. Hollatz and J. Fischer. Ber. dtsch. chem. Ges. 65, 535 (1932).

# Potassium Hydroxyfluoroborate KBF<sub>3</sub>OH

 $\begin{array}{r} 2 \text{ KHF}_2 + \text{H}_3 \text{BO}_3 = \text{KBF}_3 \text{OH} + \text{KF} + 7 \text{ H}_2 \text{O} \\ 156.22 \quad 61.84 \quad 123.96 \quad 58.1 \quad 36 \end{array}$ 

Technical grade  $\text{KHF}_2$  (100 g.) is dissolved in 250 ml. of water contained in a polyethylene beaker. The  $K_2 \text{SiF}_6$  and the undissolved  $\text{KHF}_2$  are filtered off after several hours of standing; the clear solution is placed in an ice-cold water bath and 40 g. of boric acid is added with stirring. Rapid dissolution occurs. Small crystals separate from the solution within an hour. They are suctionfiltered on a fritted glass filter, washed with a small amount of ice-cold water and with 95% methanol solution and acetone. The salt is then dried at 120°C.

#### PROPERTIES:

Melts without decomposition. Less soluble in water than  $KBF_4$ . Yields no precipitate with nitron acetate; hydrolyzed by KOH more readily than  $\text{KBF}_{4}$ . Recrystallizable from water without decomposition.

REFERENCE:

C. A. Wamser. J. Amer. Chem. Soc. 70, 1209 (1948).

## Nitrosyl Fluoroborate

## NOBF<sub>4</sub>

 $\begin{array}{rrr} 2\,\text{HBF}_4 \,+\, N_2\text{O}_3 = \,2\,\text{NOBF}_4 \,+\, \text{H}_2\text{O} \\ 175.65 & 76.01 & 233.65 & 18.01 \end{array}$ 

Dry  $N_2O_3$  (prepared by the action of concentrated nitric acid on  $As_2O_3$ ) is introduced into a platinum dish containing highly concentrated fluoroboric acid (see p. 221) until the dish contents thicken almost completely to a thick slurry and no longer absorb  $N_2O_3$ . The translucent crystalline slurry is suction-filtered on a platinum filter crucible and the remaining liquor separated by pressing. The mother liquor is concentrated in the platinum dish until the appearance of a pronounced white vapor, following which more  $N_2O_3$  is introduced. In this way additional crystals are obtained.

The suction-filtered NOBF<sub>4</sub> · H<sub>2</sub>O is vacuum-dried over  $P_2O_5$ for two days. It is then transferred to a thick-wall glass tube, where it is sealed off under liquefied N<sub>2</sub>O<sub>3</sub> at -150°C. After several hours of standing the tube is opened and the excess N<sub>2</sub>O<sub>3</sub> is allowed to escape. Finally, the preparation is left standing under vacuum and over  $P_2O_5$  and CaO for a period of several days.

In order to obtain pure NOBF<sub>4</sub> the product is vacuum-sublimed at a pressure of 0.01 mm. The sublimation apparatus consists of a glass tube sealed at one end, with an inserted water-cooled cold finger. A connecting tube, emerging from the side, leads to a mercury pump. Heating to  $200-250^{\circ}$ C is effected by means of a paraffin bath. The NOBF<sub>4</sub> is collected on the cold finger as a colorless, hard, crystalline deposit and may be scraped off with a knife.

Nitrosyl fluoroborate may be stored in glass bottles. Used to prepare NOF.

PROPERTIES:

Formula weight 116.83. Colorless, birefringent, hygroscopic flakes, crystallizing in rhombic form, which are decomposed by water, releasing nitric oxides. The dry compound does not attack glass.

d 4<sup>25</sup> 2.185.

REFERENCES:

E. Wilke-Dörfurt and G. Balz. Z. anorg. allg. Chem. <u>159</u>, 219 (1927); G. Balz and E. Mailänder. Z. anorg. allg. Chem. <u>217</u>, 162 (1934).

## Aluminum Fluoride

AlF<sub>3</sub>

$$(\mathbf{NH_4})_{3}\mathbf{AlF_6} = \mathbf{AlF_3} + 3\mathbf{NH_4F} \\ 195.09 \\ 83.97 \\ 111.22$$

The  $(NH_4)_3AlF_6$ , contained in a small platinum vessel, is heated to red heat in a nitrogen stream until constant weight is attained. Dehydration of  $AlF_3 \cdot 3H_2O$  does not produce completely oxidefree  $AlF_3$ .

**PROPERTIES:** 

White powder, sparingly soluble in water, acids and alkalis; resistant even to fuming with concentrated  $H_2SO_4$  but may be hydrolyzed with steam at 300-400°C.

Solubility in water (at  $25^{\circ}$ C): 0.559 g./100 ml.

M.p. above 1260°C, subl. t. 1260°C; d2.882. Hexagonal crystals.

REFERENCE:

W. Blitz and E. Rahlfs. Z. anorg. allg. Chem. 166, 370 (1927).

AlF<sub>3</sub> • 3 H<sub>2</sub>O

$$\begin{array}{rrr} Al + 3 \, HF + 3 \, H_2O = AlF_3 \cdot 3 \, H_2O + 1^{1/2} \, H_2 \\ 27.0 & 60.0 & 54.0 & 138.0 & 3.0 \end{array}$$

Aluminum foil is added piece-by-piece to 15% HF solution contained in a platinum dish. The reaction temperature is maintained below  $25^{\circ}$ C by periodic dipping of the dish in an ice-water bath. After some time, the initial rather vigorous reaction virtually ceases even upon addition of further quantities of aluminum. The solution is filtered through a polyethylene filter into a polyethylene dish; additional pieces of aluminum foil are added to the filtrate, and the latter is allowed to crystallize in a refrigerator for 24 hours. The crystals are washed with some water and dried at room temperature on a clay plate.

If  $AlF_3 \cdot 3 H_2O$  is not crystallized at 0°C, but instead the solution is evaporated on a steam bath until crystallization begins, a second

modification of  $AlF_3 \cdot 3$  H<sub>2</sub>O is obtained. This differs from the previously described product with regard to water solubility and powder diffraction pattern.

PROPERTIES:

White crystalline compound. Drying down to the trihydrate stage may be effected only slowly. Two moles of water of crystallization can be removed on the water bath, yielding the monohydrate.

**REFERENCES:** 

- W. F. Ehret and F. J. Frere. J. Amer. Chem. Soc. <u>67</u>, 64 (1945).
- W. Fischer and E. Bock. Z. anorg. allg. Chem. 262, 54 (1950).

## Ammonium Hexafluoroaluminate

## (NH4)3A1F6

 $6 \text{ NH}_4\text{F} + \text{Al}(\text{OH})_3 = (\text{NH}_4)_3 \text{AlF}_6 + 3 \text{ NH}_4 \text{OH}$ 222.24 77.99 195.06 105.15

Freshly precipitated hydrated aluminum oxide is introduced portionwise into a hot, rather concentrated  $NH_4F$  solution. A gelatinous precipitate results which settles easily when the solution is boiled down. The supernatant liquid is decanted or suctionfiltered, and the precipitate is washed with an alcohol-water solution and dried at 105°C.

SYNONYMS:

Ammonium cryolite, ammonium aluminum fluoride.

PROPERTIES:

White, fine crystalline powder, thermally stable to over  $100^{\circ}$ C. Solubility in water: (0°) 4 g.; (25°C) 7.7 g./liter. Does not attack glass.

d 1.78. Cubic crystals.

REFERENCES:

H. v. Helmolt. Z. anorg. allg. Chem. 3, 127 (1893).

E. Petersen. J. prakt. Chem. (2) 40, 55 (1889).

226

#### Ammonium Tetrafluoroaluminate

#### NH4AlF4

 $\begin{array}{c} (\rm NH_4)_3 \rm AlF_6 = \rm NH_4 \rm AlF_4 + 2 \, \rm NH_4 F \\ 195.1 & 121.0 & 74.1 \end{array}$ 

I. Under specific conditions, the thermal decomposition of  $(NH_4)_3AIF_6$  to  $AIF_3$  and  $NH_4F$  proceeds through the intermediate stage of  $NH_4AIF_4$ . A nickel or copper boat containing  $(NH_4)_3AIF_6$  is placed in a quartz or glass tube. Dry nitrogen gas is introduced on one side, and the whole device is heated in a furnace to 300°C. The subliming  $NH_4F$  is collected either in the cooler part of the reaction tube or in a receiver attached to the latter. Pure  $NH_4AIF_4$  remains in the boat. Raising the reaction temperature above 350°C results in further decomposition to  $AIF_3$ . Moisture must be carefully excluded in all these preparations.

II. The ammonium tetrafluoroaluminate may also be produced via the wet route by precipitation of a concentrated hydrofluoric acid-AlF<sub>3</sub> solution with  $NH_3$ .

PROPERTIES:

Crystallizes in the tetragonal system and is isomorphic with  $\text{TlAlF}_{4^{\bullet}}$ 

**REFERENCES:** 

- E. Thilo. Naturwiss. <u>26</u>, 529 (1938).
- C. Brosset. Z. anorg. allg. Chem. 239, 301 (1938).

#### Gallium (III) Fluoride

GaF<sub>3</sub>

May be prepared via thermal decomposition of ammonium hexafluorogallate.

$$\begin{array}{c} (\mathrm{NH_4})_8(\mathrm{GaF_6}) = \mathrm{GaF_3} + 3\,\mathrm{NH_4F} \\ 237.84 & 126.72 & 111.12 \end{array}$$

An alundum boat containing  $(NH_4)_3$ GaF<sub>g</sub> is placed in a nickel tube and heated for several hours in a stream of F<sub>2</sub> gas at 400°C.

Dehydration of  $GaF_3 \cdot 3H_2O$ , either under vacuum or in a fluorine stream, does not produce oxide-free  $GaF_3$ .
PROPERTIES:

Colorless compound, stable to cold and hot water. In contrast to  $GaF_3 \cdot 3 H_2O$ ,  $GaF_3$  is very sparingly soluble in water. May be sublimed without decomposition in a nitrogen stream at temperatures above  $800^{\circ}C$ .

M.p. >1000°C, b.p.  $\sim$ 950°C, d  $\sim$ 3; after heating in a fluorine stream to 630°C, d 4.47. Solubility in water (room temperature) 0.0024 g./100 ml.; in hot hydrochloric acid, 0.0028 g./100 ml.

REFERENCE:

O. Hannebohn and W. Klemm. Z. anorg. allg. Chem. 229, 342 (1936).

### Ammonium Hexafluorogallate

#### (NH<sub>4</sub>)<sub>3</sub>(GaF<sub>6</sub>)

 $\begin{array}{rl} Ga(OH)_3 + 3 HF + 3 NH_4F = (NH_4)_3(GaF_6) + 3 H_2O \\ 120.74 & 60.03 & 111.12 & 237.84 & 54.05 \end{array}$ 

Two grams of Ga (OH)<sub>3</sub> are dissolved in 40% HF solution contained in a platinum dish and the solution evaporated almost to dryness. The residue is dissolved in the least possible quantity of water, and cold, saturated solution of 6 g. of  $NH_4F$  is added. The  $(NH_4)_3GaF_6$ settles out immediately in well-formed crystals.

PROPERTIES:

Colorless crystalline salt, converts to  $Ga_2O_3$  on heating in air; heating in vacuum at 220°C results in formation of GaN, proceeding through several stages. Crystallizes in octahedra.

REFERENCE:

O. Hannebohn and W. Klemm, Z. anorg. allg. Chem. 229, 341 (1936).

## Indium (III) Fluoride

InF<sub>8</sub>

I. Thermal decomposition of  $(NH_4)_3 In F_6$  in a stream of fluorine gas.

 $(\mathrm{NH}_{4})_{3}(\mathrm{InF}_{6}) = \mathrm{InF}_{3} + 3 \mathrm{NH}_{4}\mathrm{F}_{282.88} + 171.76 \mathrm{III.12}_{111.12}$ 

A small sintered corundum vessel containing  $(NH_4)_3 In F_6$  is placed in a nickel tube and heated in a fluorine stream to constant weight. п.

$$2 \operatorname{In}_{2}O_{3} + 6 \operatorname{F}_{2} = 4 \operatorname{In} \operatorname{F}_{3} + 3 \operatorname{O}_{2}$$
  
555.04 228 637.04 96

A small sintered alumina vessel containing  $In_2O_3$  is fluorinated in a quartz tube (in an apparatus similar to that used for the preparation of TlF<sub>3</sub>). After gentle initial heating, the reaction proceeds (occasionally with incandescence) without additional supply of heat. The progress of the conversion may be checked since the yellow oxide becomes colorless and an increase in volume takes place simultaneously. To obtain a completely oxide-free preparation the product must be kept for several hours at 500°C in a nickel tube, while a stream of fluorine is passed over it.

#### PROPERTIES:

Colorless compound, stable to cold and hot water; very sparingly soluble in water (in contrast to  $InF_3 \cdot 3 H_2O$ ) although readily soluble in dilute acids. Reduced to almost pure  $InF_2$  by a very slow stream of hydrogen at 300°C; a fast stream of the latter reduces it to the metal.

M.p.  $1170^{\circ}$ C, b.p. >1200°C; d 4.39. Solubility in water at room temperature: 0.040 g./100 ml.

REFERENCE:

O. Hannebohn and W. Klemm. Z. anorg. allg. Chem. 299, 342(1936).

### Ammonium Hexafluoroindate

#### $(NH_4)_3(InF_6)$

 $\begin{array}{l} In(OH)_3 + 3 HF + 3 NH_4F = (NH_4)_3(InF_6) + 3 H_2O \\ 165.78 & 60.03 & 111.12 & 282.88 & 54.03 \end{array}$ 

Two grams of  $In(OH)_3$  are dissolved in 40% HF solution contained in a polyethylene dish and concentrated almost to dryness. The residue is dissolved in the least possible amount of water, and a cold, saturated solution containing 6 g. of NH<sub>4</sub>F is added. The volume is then reduced until crystallization begins.

PROPERTIES:

Colorless substance, crystallizing as octahedra; heating in vacuum decomposes it, forming InN.

REFERENCE:

O. Hannebohn and W. Klemm. Z. anorg. allg. Chem. <u>229</u>, 342 (1936).

### Thallium (I) Fluoride

#### TlF

 $\begin{array}{c} \text{Tl}_2\text{CO}_3 + 2\,\text{HF} = 2\,\text{TlF} + \,\text{CO}_2 + \,\text{H}_2\text{O} \\ \text{468.79} \quad 40 \quad 446.78 \quad 44.0 \quad 18.01 \end{array}$ 

Thallium carbonate is dissolved in an excess of 40% HF solution and evaporated twice to dryness. The product is then melted in a platinum crucible.

May be used for the preparation of fluorine-containing esters.

PROPERTIES:

Formula weight 223.39. Yellow liquid; hard, shiny, white, nonhygroscopic crystals which deliquesce when breathed upon, but resolidify at once.

M.p.  $327^{\circ}$ C, b.p.  $655^{\circ}$ C;  $d_4^{2^{\circ}}$  8.36. Solubility in water at  $20^{\circ}$ C: 78.8 g. in 21.2 g. H<sub>2</sub>O. A concentrated aqueous solution is strongly alkaline. Sparingly soluble in alcohol. It has a rhombic (deformed rock salt) structure.

**REFERENCES:** 

J. A. A. Ketelaar. Z. Kristallogr. <u>92</u>, 30 (1935). E. Hayek. Z. anorg. allg. Chem. <u>225</u>, 47 (1935).

## Thallium (III) Fluoride

TlF<sub>3</sub>

 $\begin{array}{rrr} 2\,\text{Tl}_{2}\text{O}_{3}\,+\,6\,\text{F}_{2}\,=\,4\,\text{Tl}\text{F}_{3}\,+\,3\,\text{O}_{2}\\ 913.56\,&228\,&1045.56\,&96 \end{array}$ 

The fluorination of  $Tl_2O_3$  is accomplished in an apparatus (see Fig. 134) consisting of a quartz reaction tube containing a quartz boat with the reagent. The fluorine gas is introduced via a 3-m.-long copper capillary which permits rotation of the reaction tube through a 90° angle. The reaction begins even at room temperature. The chocolate-brown  $Tl_2O_3$  changes color, going through black to



Fig. 134. Preparation of thallium (III) fluoride.

brown-red. The product finally becomes pure white. Fluorination should proceed very slowly, since otherwise the product fuses into a yellowish mass and not all of the material reacts. Toward the end of the fluorination the temperature is increased to 300°C.

This apparatus is suitable for all fluorinations involving elemental fluorine where the product is a nonvolatile fluoride (CuF<sub>2</sub>, AgF<sub>2</sub>, CeF<sub>4</sub>, CoF<sub>3</sub>, GaF<sub>3</sub>, InF<sub>3</sub>).

As soon as the reaction is completed, the drying tube is removed and a quartz tube with an ampoule is attached (see Fig. 134). The reaction tube is now rotated  $90^{\circ}$  and the preparation is poured into the quartz ampoule while maintaining a fluorine stream. The TlF<sub>3</sub>, sealed in the quartz ampoule in this way, can be preserved for a long period of time.

PROPERTIES:

Formula weight 261.39. White substance, very sensitive to moisture, reacts instantaneously with water, forming a black precipitate. Heating  $TlF_3$  in air causes decomposition, but it can be melted in a fluorine atmosphere.

M.p. 550°C, b.p. >550°C; d<sup>25</sup><sub>4</sub> 8.36.

REFERENCE:

O. Hannebohn and W. Klemm. Z. anorg. allg. Chem. 229, 343 (1936).

#### **Beryllium Fluoride**

BeF<sub>2</sub>

 $(\mathbf{NH_4})_2 \mathbf{BeF_4} = \mathbf{BeF_2} + 2 \mathbf{NH_4F}$ 121.10 47,02 74.08

Ammonium tetrafluoroberyllate (see next preparation) is placed in a Pt boat and heated to a red glow, excluding atmospheric moisture as far as possible. Ammonium fluoride sublimes, and the  $BeF_2$  remains in the boat in the form of a translucent glass.

PROPERTIES:

Colorless, very hygroscopic, soluble in water in all proportions, insoluble in anhydrous HF, sparingly soluble in absolute alcohol, considerably more soluble in 90% alcohol, appreciably soluble in alcohol-ether solution. Volatilizes noticeably at 800°C.

M.p.  $800^{\circ}$ C (melts in the same manner as glass, that is, with preliminary softening).

d<sup>25</sup> 1.986. Tetragonal structure.

**REFERENCE:** 

P. Lebeau. Comptes Rendus Hebd. Séances Acad. Sci. <u>126</u>, 1418 (1898).

## Ammonium Tetrafluoroberyllate

#### $(NH_4)_2BeF_4$

 $\begin{array}{rrrr} 4 \ \mathrm{NH_{4}F} \ + \ \mathrm{Be(OH)_{2}} = (\mathrm{NH_{4})_{2}BeF_{4}} \ + \ 2 \ \mathrm{NH_{4}OH} \\ 148.16 & 43.04 & 121.10 & 70.10 \end{array}$ 

Beryllium hydroxide is introduced portionwise into hot  $NH_4F$  solution. Concentration and cooling of the nearly saturated, clear solution leads to very rapid precipitation of small, colorless needles and prisms. They are suction-filtered, washed with some dilute alcohol, and dried at  $105^{\circ}C$ .

PROPERTIES:

Colorless crystals, decrepitate on heating, with subsequent melting and evolution of NH<sub>4</sub>F. Crystallizes in rhombic bipyramidal form.

REFERENCE:

H. v. Helmolt. Z. anorg. allg. Chem. 3, 129 (1893).

## **Magnesium Fluoride**

MgF<sub>2</sub>

Magnesium carbonate is dissolved in an excess of 40% HF solution contained in a platinum dish; the solution is concentrated

to dryness and dried in vacuum at  $150^{\circ}$ C. In order to obtain coarse MgF<sub>2</sub> crystals, the product is heated together with NH<sub>4</sub>F. Magnesium fluoride may be stored in glass containers.

PROPERTIES:

Colorless compound, very slightly soluble in water. Solubility (18°C) 0.087 g./liter.

M.p. 1248°C, b.p. 2260°C; d 3.148. Hardness: 6 (Mohs). Rutile structure.

REFERENCE:

W. Klemm, W. Tilk and S. von Müllenheim. Z. anorg. allg. Chem. <u>176</u>, 13 (1928); private communication from the Institute of Inorganic Chemistry of the University of Münster, unpublished.

#### **Calcium Fluoride**

CaF<sub>2</sub>

$$\begin{array}{c} CaCO_3 \,+\, 2\,HF \,=\, CaF_2 \,+\, CO_2 \,+\, H_2O \\ 100.07 \qquad 40 \qquad 78.08 \qquad 44 \qquad 18.01 \end{array}$$

Hydrofluoric acid (40%) is added with constant agitation to a suspension of 100 g. of  $CaCO_3$  in 100 ml. of boiling water contained in a large polyethylene dish. The addition is continued until evolution of CO<sub>2</sub> gas almost ceases. The mixture is filtered hot, and the precipitate on the filter is treated with dilute acetic acid until all effervescence stops. It is then thoroughly washed with hot water and finally dried at  $300^{\circ}C$ .

Fluorine ions precipitated with  $Ca^{2+}$  ions in the absence of carbonate produce gelatinous  $CaF_2$ , which is difficult to filter and wash.

May be used for manufacture of fluorsparapparatus (see p. 152).

Repeated treatment of natural fluorspar powder with concentrated hydrochloric and hydrofluoric acids results in almost pure crystalline  $CaF_2$ , which nevertheless is not well suited for making fired fluorspar vessels.

#### PROPERTIES:

White powder. Solubility in water at 18°C: 0.015 g./liter; soluble to some extent in mineral acids.

M.p. 1418°C, b.p. 2500°C; d. 3.18. Cubic (fluorite) structure. ture.

**REFERENCE:** 

O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Springer Verlag, Berlin, 1920, p. 89.

#### Strontium Fluoride

#### SrF<sub>2</sub>

 $SrCO_3 + H_2F_2 = SrF_2 + CO_2 + H_2O_147.64 40 125.63 44 18.01$ 

Strontium carbonate is dissolved in an excess of 40% hydrofluoric acid solution contained in a platinum dish. The solution is evaporated to dryness on a hot plate and dehydrated under vacuum at  $150^{\circ}$ C.

Strontium fluoride is stored in glass containers.

PROPERTIES:

Colorless powder. Solubility in water (18°C) 0.117 g./liter. M.p. 1190°C, b.p. 2460°C; d. 2.44. Cubic (fluorite) structure.

REFERENCE:

J. J. Berzelius. Pogg. Ann. 1, 20 (1824).

## **Barium Fluoride**

BaF₂

 $\begin{array}{rrr} BaCO_3 + 2 HF = BaF_2 + CO_2 + H_2O \\ 197.37 & 40 & 175.36 & 44 & 18.01 \end{array}$ 

Barium carbonate is dissolved in an excess of 40% HF solution contained in a platinum dish. The solution is evaporated to dryness and the residue heated to a red glow.

The substance is stored in glass containers.

PROPERTIES:

Colorless, transparent, small crystals. Solubility in water (18°C) 1.6 g./liter. Soluble in HF and NH<sub>2</sub>Cl solutions.

234

M.p. 1353°C, b.p. 2260°C; d 4.83. Cubic (fluorite) structure.

REFERENCE:

W. Olbrich. Thesis, Technische Hochschule, Breslau, 1929, p. 2.

#### Lithium Fluoride

#### LiF

 $\begin{array}{c} \text{Li}_2\text{CO}_3 + 2\,\text{HF} = 2\,\text{LiF} + \text{CO}_2 + \text{H}_2\text{O} \\ 73.88 \quad 40 \quad 51.88 \quad 44 \quad 18.01 \end{array}$ 

Lithium carbonate is added to 40% HF solution contained in a platinum dish. The mixture is evaporated to dryness, thoroughly calcined, pulverized with a platinum pestle and stored in paraffin bottles.

Lithium fluoride may be used in the preparation of single crystals for optical, photoelectric and dielectric studies, as well as for coating crucibles used in melting Li metal.

PROPERTIES:

Formula weight 25.94. White, granular powder. Solubility in water  $(18^{\circ}C) 0.27 \text{ g.}/100 \text{ ml.}$ 

M.p. 842°C, b.p. 1676°C. Volatilizes between 1100 and 1200°C; d. (solid)(20°C) 2.640, d. (liq.) (1058°C) 1.699. Cubic (rock salt) structure.

**REFERENCE:** 

H. von Wartenberg and H. Schulz. Z. Elektrochem. 27, 568 (1921).

## Sodium Fluoride

NaF

 $\begin{array}{r} \text{NaOH} + \text{HF} = \text{NaF} + \text{H}_2\text{O} \\ 40 \quad 20 \quad 42 \quad 18 \end{array}$ 

The stoichiometric quantity of NaOH or Na<sub>2</sub>CO<sub>3</sub> is added to 40% HF solution contained in a polyethylene dish. Sodium fluoride precipitates out at once; it is suction-filtered and dried in an oven at  $110^{\circ}$ C.

Dry NaF may be stored in glass containers.

PROPERTIES:

White powder. Solubility in water  $(15^{\circ}C)$  4 g.;  $(25^{\circ}C)$  4.3 g./100 ml.; insoluble in alcohol.

M.p. 993°C, b.p. 1704°C; d 2.78. Cubic (rock salt) structure.

**REFERENCE:** 

A. E. Müller. Chem. Ztg. 52, 5 (1928).

#### **Potassium Fluoride**

KF

I.

II.

 $KHF_2 = KF + HF$ 78.11 58.11 20

Thermal decomposition of  $\rm KHF_2$  yields the purest KF. To obtain this,  $\rm KHF_2$  contained in a platinum dish is heated in an electric furnace to 500°C (under a hood). A platinum funnel is placed over the dish and well-dried nitrogen is introduced through the funnel stem.

 $\begin{array}{c} \mathrm{KF} \cdot 2\,\mathrm{H_2O} = \mathrm{KF} + 2\,\mathrm{H_2O} \\ \mathrm{94.13} \quad 58.11 \quad 36.02 \end{array}$ 

The stoichiometric quantity of chlorine-free potassium hydroxide (or  $K_2CO_3$  solution) is introduced into a polyethylene dish containing 40% HF solution. The KF  $\cdot$  2 H<sub>2</sub>O separates out as a crystalline slurry on cooling. The latter is suction-filtered in polyethylene equipment, pressed between filter paper sheets and dried as much as possible without melting in a vacuum drying oven (m.p. 46°C).

PROPERTIES:

White, hygroscopic, deliquescent powder. Solubility in water  $(18^{\circ}C)$  92.3 g./100 ml.; insoluble in alcohol.

M.p.  $857^{\circ}$ C, b.p.  $1503^{\circ}$ C; d 2.48. Cubic (rock salt) structure.

REFERENCE:

E. Lange and A. Eichler. Z. phys. Chem. 129, 286 (1927).

# Potassium Hydrogen Fluoride

#### KF·HF

 $\begin{array}{c} \text{KOH} + 2\,\text{HF} = \text{KHF}_2 + \text{H}_2\text{O} \\ 56.11 \quad 40 \quad 78.11 \quad 18.01 \end{array}$ 

The stoichiometric quantity of chlorine-free potassium hydroxide (or  $K_2CO_3$  solution) is introduced into an ice-cooled Pt, Ag or Ni dish containing 40% HF solution. The KHF<sub>2</sub> precipitates out and can be suction-filtered at once. It can be recrystallized from hot water. It is dried at 120-150°C in a stream of completely dry air.

To produce absolutely anhydrous  $\text{KHF}_2$ , the precipitate is treated with fluorine gas in a cylindrical iron or copper vessel provided with a bottom tube through which fluorine gas can be introduced. The vessel cover is equipped with a gas outlet. The drying process is complete when fluorine gas is detected at the outlet.

The product may be stored in aluminum cans; large quantities of the substance are kept in wooden drums.

It is used in the preparation of fluorine gas and pure KF.

SYNONYM:

Potassium bifluoride.

PROPERTIES:

Colorless salt, readily soluble in water. M.p. 239°C; d 2.37. Tetragonal structure.

REFERENCE:

E. Lange and A. Eichler. Z. phys. Chem. 129, 285 (1927).

#### Potassium Tetrafluorobromate (III)

#### KBrF4

 $3 \text{ KCl} + 4 \text{ BrF}_3 = 3 \text{ KBrF}_4 + \text{Br} + 3 \text{ Cl}$ 223.68 547.64 525.0 79.91 106.41

A large excess of  $BrF_{g}$  is slowly (dropwise) added to about 0.5 g. of KCl contained in a quartz vessel. The mixture is then kept for several minutes at 20°C and then rapidly cooled. The quartz container is then connected to a quartz trap immersed in liquid nitrogen, which in turn is connected to a vacuum pump. The excess  $BrF_{3}$ is vacuum distilled into the quartz trap. PROPERTIES:

White, crystalline powder; decomposes on heating, with elimination of  $BrF_3$ . Reacts rapidly with water (decomposition), but less vigorously than  $BrF_3$ . Stable to  $CCl_4$ , acetone and dioxane. Attacks platinum metal when heated.

REFERENCE:

A. G. Sharpe and H. J. Emeléus. J. Chem. Soc. (London) 1948, 2136.

#### Potassium Hexafluoroiodate (V)

KI F<sub>6</sub>

 $KF + IF_5 = KIF_6$ 58.11 174.91 233.02

Potassium fluoride is dissolved in boiling iodine (V) fluoride contained in a quartz vessel. The solubility is 1 g. of KF per 100 g. of IF<sub>5</sub>. The KIF<sub>6</sub> precipitates out as white crystals when the solution is cooled. The excess iodine (V) fluoride is removed by evaporation at  $15-20^{\circ}$ C and a pressure of 2-5 mm.

**PROPERTIES:** 

White crystals, slightly soluble in cold, but more readily soluble in hot iodine (V) fluoride. Decomposes when heated to  $200^{\circ}$ C; hydrolyzed by water with evolution of heat; stable to CCl<sub>4</sub>.

M.p. about 200°C.

**REFERENCE:** 

H. J. Emeleus and A. G. Sharpe. J. Chem. Soc. (London) 1949, 2206.

## Copper (II) Fluoride

CuF<sub>2</sub>

Т	
T	٠

$$CuCl_2 + F_2 = CuF_2 + Cl_2$$
  
134.48 38 101.57 70.92

Anhydrous CuCL contained in a copper boat is fluorinated with  $F_2$  or ClF<sub>3</sub> at 400°C in the apparatus already described for the preparation of TlF<sub>3</sub> (see p. 231).

Π.

$$\begin{array}{c} \text{CuO} + 2\,\text{HF} = \text{CuF}_2 + \text{H}_2\text{O} \\ \text{79.57} & 40 & 101.57 & 18.01 \end{array}$$

Copper (II) oxide is dissolved in an excess of 40% hydrofluoric acid solution contained in a polyethylene dish, so as to form solid  $CuF_2 \cdot 5 H_2O \cdot 5$  HF. This is then transferred to a small platinum boat, which is inserted in a copper or nickel tube. The salt is dehydrated at 400°C in a completely dry HF stream (see Fig. 141, p. 267). The excess HF is displaced by a stream of nitrogen. The product is cooled under a nitrogen blanket.

The product is stored in sealed glass ampoules.

**PROPERTIES:** 

White, crystalline powder, sensitive to air, sparingly soluble in cold water, hydrolytically cleaved by hot water. Solubility in water ( $20^{\circ}$ C) 4.7 g./100 ml.

M.p. 950°C.

**REFERENCES:** 

P. Henkel and W. Klemm. Z. anorg. allg. Chem. 222, 74 (1935);
H. von Wartenberg. Z. anorg. allg. Chem. 241, 381 (1939).

## Silver Subfluoride

Ag<sub>2</sub>F

Prepared by cathodic reduction of silver fluoride solution:

 $AgF + Ag = Ag_2F$ 126.88 107.88 234.76

Silver carbonate is added to warm, pure 40% hydrofluoric acid solution until no more dissolves. After addition of 2 g. of  $NH_4F$ , the undissolved material is filtered off in the dark.

A platinum dish serves as the electrolysis vessel and the cathode. It is placed on a water bath at  $50^{\circ}$ C. A 100-g. solid Ag bar with a welded-on Ag lead wire is used as the anode. The maximum current density of the cathode is 0.002 amp./cm? The voltage drop across the electrodes is 1.4 v. A 6-v. battery is used as the power supply; the current is 0.07-0.1 amp.

Under these conditions, 15-20 g. of large, greenish, shiny crystals is produced in 48 hours. Occasionally Ag precipitates out instead of Ag<sub>2</sub>F at the start of the reaction. Since during electrolysis Ag passes into solution, the silver concentration of the solution remains constant. Following electrolysis the crystals are separated from the electrolyte by decantation. They are freed from the adhering AgF solution by pressing between filter paper and are stored in a desiccator.

PROPERTIES:

Large, shiny, bronze-colored, greenishly opalescent crystals which slowly turn gray-black on exposure to light. On heating to  $150^{\circ}$ C Ag<sub>2</sub>F turns gray; at 700°C it disproportionates quantitatively into AgF + Ag. Decomposes in water to gray Ag powder. Stable to alcohol.

d. 8.57. Hexagonal structure.

REFERENCES:

A. Hettich. Z. anorg. allg. Chem. <u>167</u>, 67 (1927).
R. Scholder and K. Traulsen. Z. anorg. allg. Chem. 197, 57 (1931).

## Silver Fluoride

#### AgF

I.

Ag <sub>2</sub> CO <sub>3</sub>	$+2 \mathrm{HF} =$	2 AgF	$+ CO_2$	$+ H_2O$
275.77	40	253.76	44.0	18.01

Coarse-grained  $Ag_2CO_3$  is prepared by precipitation from  $AgNO_3$  solution with dilute NaHCO<sub>3</sub> or Na<sub>2</sub>CO<sub>3</sub> solution. The precipitate is purified by washing until the test for nitrate ion is negative.

The  $Ag_2CO_3$  thus obtained is dissolved in an excess of 40% hydrofluoric acid solution contained in a platinum dish; the clear solution is rapidly evaporated on an open flame until the beginning of crystallization. It is then evaporated to dryness on a sand bath (constant agitation with a platinum spatula; rubber gloves must be worn). The fine AgF produced is brown-black (contains Ag<sub>2</sub>O and Ag). May be used for fluorination without further purification.

To prepare very pure AgF, anhydrous HF is passed over coarsegrained Ag<sub>2</sub>CO<sub>3</sub> contained in a platinum tube the temperature of which is gradually raised to  $300^{\circ}$ C. The apparatus used is identical to the one used for the preparation of CoF<sub>2</sub> (Fig. 141, p. 267). After cooling in a stream of dry nitrogen, the pure, dry product is easily poured from the platinum tube. The yield is quantitative. II. Pure, crystalline anhydrous AgF can be more conveniently obtained via electrolysis of a solution of KF in acetic acid, using silver anodes. A 7% solution of KF in glacial acetic acid is electrolyzed in a vessel containing an Ag ingot or bar as the anode and a platinum gauze cathode. The current must be greater than 40 ma. Under these conditions, the AgF formed at the anode falls off and collects at the bottom of the electrolytic bath. The product is filtered, washed consecutively with glacial acetic acid and anhydrous benzene, and placed in a vacuum desiccator at room temperature to remove the adhering benzene. The yield is 99.5%. At 120 ma. and 20 v., 0.5 g. of AgF is obtained in 60 minutes. III. Other preparative methods: thermal decomposition of AgBF<sub>4</sub> [A. G. Sharpe, J. Chem. Soc. (London) 1952, 4538].

Silver subfluoride is stored in opaque glass bottles. Used to fluorinate organic compounds.

PROPERTIES:

White, flaky crystals with a flexibility similar to that of horn; pulverized with difficulty, but may be hammered into plates and cut with shears. Very hygroscopic. Darkens upon exposure to light. Solubility in water ( $15^{\circ}$ C) 135 g./100 ml. Also soluble in HF, CH<sub>3</sub>COOH and CH<sub>3</sub>CN.

M.p. 435°C; d. 5.852. Cubic (rock salt) structure.

REFERENCES:

- I. O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin, 1920, p. 37; K. Fredenhagen. German Patent Application F 293 30 IV b/12 i, August 1, 1930.
- II. H. Schmidt. Z. anorg allg. Chem. 270, 196 (1952).

## Silver (II) Fluoride

AgF<sub>2</sub>

I.

 $2 \operatorname{AgCl} + 2 \operatorname{F}_2 = 2 \operatorname{AgF}_2 + \operatorname{Cl}_2$ 286.67 76 291.76 70.91

Fluorine gas is passed over a nickel boat containing AgCl. The boat is placed in a nickel tube (the apparatus is identical to that described for the preparation of  $TlF_3$ , p. 231). External cooling must be provided at the start of the reaction so that the temperature does not exceed  $80^{\circ}$ C. Otherwise a ternary mixture consisting of AgCl, AgF and AgF<sub>2</sub> is formed. This fuses, making further absorption of fluorine difficult. The temperature is then gradually increased to  $250^{\circ}$ C. The product is allowed to cool in a fluorine stream. The fluorine is then displaced with dry N<sub>2</sub>. The yield is 95%, based on AgCl.

Π.

 $Ag + F_2 = AgF_2$ 107.88 38 145.88

"Molecular" Ag is fluorinated in the apparatus described above. The reaction begins at room temperature with evolution of heat, resulting in a yellow to brown product. Careful external cooling should be provided so that the temperature does not exceed  $60^{\circ}$ C. When the reaction subsides the temperature is gradually increased to  $250^{\circ}$ C. The product is allowed to cool in a fluorine stream, which is then displaced with dry N<sub>2</sub>.

The product may be stored in sealed quartz ampoules or in iron containers. It may be used for fluorination of organic compounds as well as for the preparation of  $COF_2$ .

PROPERTIES:

White when pure; otherwise somewhat brown-tinged. Thermally stable up to 700°C; high chemical reactivity. Instantly hydrolyzed by water.

M.p. 690°C; d 4.7; AH (formation) 84.5 kcal.

**REFERENCES:** 

- I. W. S. Struve et al. Ind. Eng. Chem. 39, 353 (1947).
- II. O. Ruff and M. Giese. Z. anorg. allg. Chem. <u>219</u>, 143 (1934);
   H. von Wartenberg. Z. anorg. allg. Chem. <u>242</u>, 406 (1939).

### Zinc Fluoride

ZnF2

$$\begin{array}{rrrr} ZnCO_3 + 2 HF = ZnF_2 + CO_2 + H_2O \\ 125.39 & 40 & 103.38 & 44 & 18.01 \end{array}$$

Zinc carbonate is added to an excess of hot aqueous hydrofluoric acid. Initially, a clear solution results. Further addition of  $ZnCO_3$  causes precipitation of  $ZnF_2$  as white, opaque crystals. The mixture is then evaporated to dryness on a hot plate.

This only partially dehydrated form of  $ZnF_2$  is used for fluorination. Absolutely anhydrous  $ZnF_2$  is much less reactive and therefore less suitable.

To obtain anhydrous  $ZnF_2$ , the precipitate must be heated to  $800^{\circ}C$  with exclusion of atmospheric moisture. It is heated in the presence of  $NH_4F$  so as to produce larger crystals.

May be stored in glass bottles.

PROPERTIES:

Transparent crystalline needles, sparingly soluble in water, somewhat soluble in dilute hydrofluoric acid, soluble in hydrochloric and nitric acids and ammonia. Solubility in water:  $5 \cdot 10^{-5}$  moles/liter.

M.p. 872°C, b.p. 1500°C; d 4.84. Tetragonal (rutile) structure.

REFERENCES:

O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin, 1920, p. 36; private communication from the Institute of Inorganic Chemistry of the University of Münster, unpublished.

## Cadmium Fluoride

#### CdF<sub>2</sub>

 $CdCO_3 + 2 HF = CdF_2 + CO_2 + H_2O$ 172.42 40 150.51 44 18.01

Cadmium carbonate is added to an excess of 40% hydrofluoric acid solution contained in a platinum dish; the mixture is evaporated to dryness on at hot plate and dehydrated in vacuum at 150°C.

The product is stored in glass containers.

PROPERTIES:

Colorless compound. Solubility in water (25°C) 4.3 g./100 ml. Soluble in hydrofluoric acid and other mineral acids, insoluble in alcohol and liquid ammonia.

M.p. 1049°C, b.p. 1748°C; d 6.33. Cubic (fluorite) structure.

REFERENCE:

W. Klemm, W. Tilk and S. von Müllenheim. Z. anorg. allg. Chem. 176, 13 (1928).

### Mercury (I) Fluoride

Hg<sub>2</sub>F<sub>2</sub>

 $Hg_2CO_3 + 2 HF = Hg_2F_2 + CO_2 + H_2O$ 40 472.13 461.22 44 18

Mercury (I) nitrate (150 g.) is dissolved in a solution of about 60 ml. of dilute HNO<sub>2</sub> in about 450 ml. of water. The solution is poured in a fine stream into a vigorously agitated solution of 50 g. of KHCO<sub>3</sub> in one liter of water. Following repeated washing with  $CO_2$ -saturated water (Dry Ice added to water), it is filtered with good suction. The wet Hg<sub>2</sub>CO<sub>3</sub> is added in small portions and with constant stirring to 40% hydrofluoric acid solution contained in a platinum dish. A yellow powder of Hg<sub>2</sub>F<sub>2</sub> settles out. The addition of Hg<sub>2</sub>CO<sub>3</sub> is continued as long as CO<sub>2</sub> is vigorously evolved; the highly dilute supernatant hydrofluoric acid solution is added. The resulting mixture is evaporated to dryness on a water bath. The product is then pulverized and heated for 2-3 hours in a drying oven at 120-150°C. The product is then immediately poured into copper containers and vacuum-sealed.

The product must be prepared in the dark or at least in diffuse light.

**PROPERTIES:** 

Yellowish crystalline powder, blackens rapidly on exposure to light; more readily soluble in water (hydrolysis) than Hg<sub>2</sub>Cl<sub>2</sub>. M.p. 570°C; d (15°C) 8.73. Tetragonal structure.

REFERENCES:

- O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin, 1920, p. 34.
- A. L. Henne and M. W. Renoll. J. Amer. Chem. Soc. 60, 1060 (1938).

## Mercury (II) Fluoride

HgF<sub>2</sub>

I.

 $\begin{array}{l} HgCl_2 \,+\, F_2 = \,HgF_2 \,+\, Cl_2 \\ 271.52 \quad 38 \quad 238.61 \quad 70.92 \end{array}$ 

A horizontal copper cylinder which can be rotated like a revolving drum about its own axis (20 r.p.m.) serves as the reaction vessel. Fluorine gas is introduced through one side of the hollow axis, while the other serves as an outlet for the reaction gas (see Fig. 135).

The copper drum is filled with dry, pulverized  $HgCl_2$  and several small pieces of copper, intended to break up crust formations. An exothermic reaction begins as soon as the fluorine is introduced. The progress of the reaction is followed by withdrawal of samples.



Fig. 135. Preparation of mercury (II) fluoride.

The samples are dissolved in nitric acid and tested for chloride ion. The reaction is considered complete as soon as chloride ion is not detectable. The product is poured into sealed copper containers. The yield is 75%, based on HgCl<sub>2</sub>.

II.  $HgO + 2 HF = HgF_2 + H_2O$ 216.62 40 283.61 18.01

In an apparatus similar to that described for the preparation of  $CoF_2$  (p. 267), 11 parts by weight of HgO, contained in a small nickel boat, are fluorinated for 4.5 hours at  $380-450^{\circ}C$  with a gas mixture consisting of 30 parts by weight of anhydrous HF and 2 parts of O<sub>2</sub>.

Small amounts of  $HgF_2$  can be prepared in an apparatus similar to that described for the preparation of  $TlF_3$  (Fig. 134, p. 231).

Mercury (II) fluoride may be used as a fluorinating agent in organic chemistry.

#### PROPERTIES:

White powder, very sensitive to moisture; hydrolyzed instantly by water, yielding a yellow color.

M.p. 645°C, b.p. >650°C; d(15°C) 8.95. Cubic (fluorite) structure.

REFERENCES:

I. A. I. Henne and T. Midgley. J. Amer. Chem. Soc. <u>58</u>, 886 (1936).

II. U.S. Patent 2,757,070.

## Scandium Fluoride

ScF<sub>3</sub>

 $\begin{array}{r} Sc(OH)_3 + 3 HF = ScF_3 + 3 H_2O \\ 96.03 & 60 & 102.10 & 54.03 \end{array}$ 

Scandium oxide or hydroxide is added to 40% hydrofluoric acid contained in a polyethylene dish until saturated. It is then evaporated; the precipitate formed is filtered off and vacuum-dried at 150-180 °C.

PROPERTIES:

White powder, very sparingly soluble in water, somewhat soluble in alkali carbonate and ammonium carbonate solutions. Completely decomposed by alkali fusion. Hexagonal structure.

REFERENCE:

Gmelin-Kraut VI, 2, p. 681.

## **Yttrium Fluoride**

YF3

 $\begin{array}{l} Y(NO_8)_3 \,+\, 3\, NH_4OH \,=\, Y(OH)_8 \,+\, 3\, NH_4NO_3 \\ 274.95 \,\,\, 105.12 \,\,\, 139.95 \,\,\, 240.15 \end{array}$ 

 $Y(OH)_3 + 3 HF = YF_3 + 3 H_2O$ 139.95 60 145.92 54.03

The hydroxide is precipitated from aqueous yttrium nitrate with ammonia. The product is washed and repeatedly evaporated to dryness in a platinum dish together with aqueous hydrofluoric acid solution.

PROPERTIES:

White powder, insoluble in HF, soluble in  $H_2SO_4$  d.4.01. Cubic structure.

REFERENCES:

E. Zintl and A. Udgard. Z. anorg. allg. Chem. <u>240</u>, 152 (1939). W. Nowacki. Z. Kristallogr.(A)<u>100</u>,242 (1939).

## Lanthanum Fluoride

LaF<sub>8</sub>

 $LaCl_3 + 3 HF = LaF_3 + 3 HCl$ 245.29 60 195.92 109.38

A hydrochloric acid solution of  $LaCl_3$  contained in a polyethylene dish is treated with 40% hydrofluoric acid; the excess HF is decanted and the residue is evaporated to dryness. PROPERTIES:

Colorless solid, insoluble in water. Hexagonal (tysonite) structure.

**REFERENCE:** 

G. P. Drossbach. Thesis, Technische Hochschule, Munich, 1905, p. 9.

## Cerium (III) Fluoride

CeF<sub>3</sub>

 $\begin{array}{rrrr} 4\,\text{CeO}_2 \,+\, 12\,\text{HF} = 4\,\text{CeF}_3 \,+\, 6\,\text{H}_2\text{O} \,+\, \text{O}_2 \\ 688.49 & 240.12 & 788.52 & 108.09 & 32.00 \end{array}$ 

A mixture of  $CeO_2$  and an excess of hydrofluoric acid is evaporated to dryness in a polyethylene dish.

PROPERTIES:

Formula weight 197.13. Colorless, powdery product. M.p. 1460°C; d 6.16.

**REFERENCE:** 

H. von Wartenberg. Z. anorg. allg. Chem. 244, 343 (1940).

## Cerium (IV) Fluoride

CeF₄

 $\begin{array}{r} 2 \, \mathrm{CeF_3} + \mathrm{F_2} = 2 \, \mathrm{CeF_4} \\ 394.26 & 38 & 432.26 \end{array}$ 

In an apparatus similar to that described for the preparation of  $TlF_3$  (p. 231), CeF<sub>3</sub> is fluorinated in a sintered alumina vessel at 500°C.

**PROPERTIES:** 

Formula weight 216.13. White, fine, crystalline salt, insoluble in water; hydrolyzes very slowly in cold water.

M.p. >650°C; d 4.77. Can be reduced to  $CeF_3$  with hydrogen at 300°C.

REFERENCES:

H. von Wartenberg. Z. anorg. allg. Chem. <u>244</u>, 343 (1940). W. Klemm and P. Henkel. Z. anorg. allg. Chem. 220, 181 (1934).

### Europium (II) Fluoride

EuF2

 $\begin{array}{l} EuF_3 + \frac{1}{2}H_2 = EuF_2 + HF\\ 208.9 & 1.0 & 189.9 & 20.0 \end{array}$ 

A small platinum vessel containing  $EuF_3$  (the preparation is the same as that of  $LaF_3$  or  $CeF_3$ ) is placed in a 20-cm.-long platinum tube, which in turn is fitted quite exactly into a quartz tube. It is heated rapidly to 900°C in a high-velocity stream of carefully purified hydrogen and then reduced at 1100°C over a period of three hours.

PROPERTIES:

Light yellow solid; C1-type structure (fluorite).

REFERENCES:

W. Klemm and W. Döll, Z. anorg. allg. Chem. <u>241</u>, 234 (1939). G. Beck and W. Nowacki. Naturwiss. 27, 495 (1938).

#### Titanium (III) Fluoride

TiF<sub>8</sub>

Ti (as the hydride) +  $3 \text{ HF} = \text{TiF}_3 + \frac{1}{2} \text{ H}_2$ 47.9 60 104.9 3

Titanium metal is hydrogenated at  $600-700^{\circ}$ C (see section on Titanium). It is then placed in a small nickel boat s, which in turn is inserted into the horizontal nickel tube a (closed at one end), and the hydride is fluorinated with gaseous HF (see Fig. 136). The open end of the tube has a cooling jacket and is sealed with picein to a copper cover b. Two copper tubes are silver-soldered into the cover and serve as inlet and outlet for the hydrogen; in addition,



Fig. 136. Preparation of titanium (III) fluoride. a) Nickel tube; b) copper cover; c) small nickel boat.



Fig. 137. Sublimation of titanium (III) fluoride. C) Quartz tube; f cold finger; e) Ni crucible; O) tubular furnace.

the inlet tube contains a concentric silver tube for the introduction of HF. The output gases pass through an empty polyethylene bottle, a bubble counter filled with paraffin oil and a polyethylene trap to freeze out excess HF gas. The closed end of the reaction tube is placed in a tubular furnace. The temperature is measured externally.

After thorough flushing of the apparatus with  $H_2$  there follows a fluorination with a 1 : 4 mixture of  $H_2$ ; HF for a period of four to five hours. The HF flow is first started at a temperature above 200°C. The temperature of the water in the cooling jacket should be higher than 20°C to prevent condensation of the HF. After

completion of the fluorination the product is left to cool in a stream of hydrogen. The yield is 90%.

A nickel crucible e containing the product (5 g.) is placed at the closed end of a quartz tube c, which is then inserted into the oblique tubular furnace o. The open end of the tube c is closed off with an adapter provided with outlets to a vacuum pump. A water-cooled copper cold finger f is sealed into the cap with picein. The cold finger terminates in a copper rod on which the TiF<sub>3</sub> crystals grow. The TiF<sub>3</sub> begins to sublime at  $10^{-2}$  to  $10^{-3}$  mm. About 80% of the product, in the form of bright blue crystals, collects on the cold finger over a period of four hours at  $1000^{\circ}$ C. A grayblack residue remains in the crucible. The TiF<sub>3</sub> thus obtained is so pure that it can be used directly for magnetic measurements.

**PROPERTIES:** 

Blue, rhombic crystals, stable in air, unusually resistant to acids and bases. Sublimation begins at about  $930^{\circ}$ C in a vacuum of less than 0.1 mm.

M.p. >1100°C;  $d_4^{25}$  2.98. Insoluble in water and alcohol. Disproportionation to TiF<sub>4</sub> and Ti begins at 950°C.

**REFERENCE:** 

P. Ehrlich and G. Pietzka. Z. anorg. allg. Chem. 275, 121 (1954).

## Titanium (IV) Fluoride

TiF₄

 $TiCl_4 + 4 HF = TiF_4 + 4 HCl$ 189.74 80 129.9 145.84

A copper or platinum Erlenmeyer flask with a detachable distillation head serves as the reaction vessel. A copper drying tube containing  $CaCl_2$  is either attached directly at the outlet of the head or after a descending condenser.

A weighed quantity of ice-cold anhydrous HF is poured into the flask, which is cooled by an ice-salt mixture (the reaction should be carried out under a hood). One half of the  $TiCl_4$ , (calculated from the above equation) is weighed into a test tube and added dropwise to the HF solution. Each drop causes a vigorous reaction and evolution of HCl gas. The mixture contained in the flask is left standing for several hours until all the ice has melted (the head and drying tube are attached). The Erlenmeyer flask is then transferred to an oil bath; the drying tube is replaced with a copper condenser attached to a lead receiver. The temperature of the oil bath is gradually raised to  $200^{\circ}$ C, as a result of which HClcontaining hydrofluoric acid distills over. The oil bath is then removed, the condenser is taken off, and the TiF<sub>4</sub> is sublimed by heating with an open Bunsen burner flame. The sublimate flows into a copper receiver, which can be sealed and which doubles as a storage container for the product. The receiver is placed over the neck of the retort and cooled with water flowing through lead coils. The head must always be warm during this procedure to prevent plugging of the equipment. The yield is 90%, based on TiCl<sub>4</sub>.

The product is stored in tightly sealed copper or iron containers.

PROPERTIES:

Colorless, loose powder; very hygroscopic, reacts with water with effervescence; dissolves in alcohol with evolution of heat; insoluble in ether.

M.p. >400°C (under pressure), subl. t. 284°C; d. (20°C) 2.798.

**REFERENCES:** 

- O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin, 1920, p. 48;
- O. Ruff and R. Ipsen. Ber. dtsch. chem. Ges. 36, 1777 (1903);
- O. Ruff and W. Plato. Ber. dtsch. Chem. Ges. 37, 673 (1904).

## Zirconium (IV) Fluoride

ZrF4

 $ZrCl_4 + 4 HF = ZrF_4 + 4 HCl$ 233.06 80 167.22 145.84

In a procedure similar to that described for the preparation of  $\text{TiF}_4$  (see above), 50 g. of  $\text{ZrCl}_4$  is gradually added to 120-150 g. of anhydrous HF. Further treatment is, however, simpler in this case since there is no necessity for subliming the  $\text{ZrF}_4$ . After the HF is distilled off, the Erlenmeyer flask is heated until the bottom is red hot. The  $\text{ZrF}_4$  is then pure, and after cooling can be stored immediately in sealable copper containers.

Other preparative method: Thermal decomposition of  $(NH_4)_2 ZrF_6$ .

PROPERTIES:

White, highly refractive translucent substance. Solubility in water: 1.32 g./100 ml. Hydrolyzed by water above 50°C. Subl. t. >600°C; d (20°C) 4.6. Monoclinic crystals. **REFERENCES:** 

- O. Ruff. Die Chemie des Fluors [Fluorine Chemistry], Berlin, 1920, p. 49.
- L. Wolter. Chem. Ztg. 51, 607 (1908).

# Vanadium (III) Fluoride

VF<sub>3</sub>

 $VCl_3 + 3 HF = VF_3 + 3 HCl$ 157.32 60 107.95 109.37

In an apparatus similar to that described for the preparation of  $\text{Co} F_2$  (p. 267), 4 g. of VCl<sub>3</sub> is treated with anhydrous HF in a nickel, sintered alumina or platinum vessel. At the start of the experiment the apparatus is flushed with dry nitrogen to displace atmospheric oxygen. During the fluorination the tube is slowly heated to 200°C. After 1.5 hours the temperature is raised to red heat. The reaction is complete when the exit gas no longer contains any HCl. The product is allowed to cool to 100°C in a stream of HF gas, after which the cooling is continued in a nitrogen stream. The yield is 95%, based on VCl<sub>3</sub>.

**PROPERTIES:** 

Yellowish-green powder, almost insoluble in water, alcohol, acetone, ethyl acetate, acetic anhydride, glacial acetic acid, toluene, CCl<sub>4</sub>, CHCl<sub>3</sub> and CS<sub>2</sub>. Becomes black in sodium hydroxide solution. M.p. >800°C; sublimation occurs at bright red heat; d 3.363.

REFERENCE:

O. Ruff and H. Lickfett. Ber. dtsch. chem. Ges. 44, 2539 (1911).

## Vanadium (IV) Fluoride

VF4

 $VCl_4 + 4 HF = VF_4 + 4 HCl$ 192.79 80 126.95 145.84

Freshly distilled, Dry Ice-cooled VCl<sub>4</sub> (40 g.) is added to 130 g. of similarly cooled anhydrous HF contained in a reactor identical to that described for the preparation of  $TiF_4$  (p. 250). A copper

reflux condenser charged with a cooling mixture is then attached to the vessel, and the reaction mixture is allowed to warm slowly to 0°C. The mixture is then boiled for several hours until no further HCl evolves. The reflux condenser is then replaced by a descending condenser and the HF is distilled off. The VF<sub>4</sub> remaining in the reactor is freed of any traces of HF by passage of a dry stream of nitrogen at 50°C. The yield is 97%, based on VCl<sub>4</sub>.

The product may be stored in sealed iron or copper containers.

PROPERTIES:

Brownish-yellow, loosely packed powder; very hygroscopic, deliquesces in air to a blue liquid; readily soluble in water, imparting a blue color to the solution. Soluble in acetone and glacial acetic acid, giving a deep green and blue-green color respectively. Only very slightly soluble in  $SO_2Cl_2$ , alcohol and chloroform. Not volatile, but disproportionates above  $325^{\circ}C$  to  $VF_3$  and  $VF_5$ . d.  $(23^{\circ}C)$  2.975.

**REFERENCE:** 

O. Ruff and H. Lickfett. Ber. dtsch. chem. Ges. 44, 2539 (1911).

## Vanadium (V) Fluoride

VF<sub>5</sub>

 $\begin{array}{c} 2 \, \mathrm{VF_4} = \mathrm{VF_5} + \mathrm{VF_3} \\ 253.9 \quad 145.95 \quad 107.95 \end{array}$ 

A small nickel or platinum boat containing VF<sub>4</sub> is placed in a nickel tube and gradually heated to  $650^{\circ}$ C in a stream of dry N<sub>2</sub>. The exit gases are collected in a large-diameter quartz trap maintained at  $-78^{\circ}$ C. The trap is attached to a drying tube with anhydrous KF to exclude atmospheric moisture. The heating process must be effected slowly, since otherwise the unreacted VF<sub>4</sub> is blown out of the reaction tube. The product is allowed to cool in a nitrogen stream and the VF<sub>5</sub> is discharged directly from the gas trap into a storage container. A greenish-yellowish residue (VF<sub>4</sub>) remains in the boat. The yield is almost quantitative.

The product is stored in sealed iron, nickel, copper or platinum containers.

PROPERTIES:

Compact, white substance, displaying a noticeable vapor pressure at room temperature, becomes yellow in air; soluble in water, W. KWASNIK

to which it imparts a yellow-red color. Readily soluble in alcohol, chloroform, acetone and ligroin; insoluble in  $CS_2$ . Decomposes toluene and ether. Slowly attacks glass at room temperature. M.p. >200°C (under pressure); subl. t. 111.2°C; d (19°C) 2.177.

REFERENCE:

O. Ruff and H. Lickfett. Ber. dtsch. chem. Ges. 44, 2548 (1911).

## Niobium (V) Fluoride

#### NbF₅

I.

 $NbCl_{5} + 5 HF = NbF_{5} + 5 HCl$ 270.20 100 187.91 182.30

In a manner similar to that described for the preparation of  $\text{TiF}_4$  (p. 250), NbCl<sub>5</sub> is introduced into twice the theoretical amount of anhydrous HF. A copper or iron reflux condenser charged with a cooling mixture is then attached to the reactor; the product is boiled for several hours with the excess HF until evolution of HCl ceases. The HF is then distilled through a downward condenser. The condenser is then replaced with a distillation head and the NbF<sub>5</sub> distilled off.

II.  $2 \text{ Nb} + 5 \text{ F}_2 = 2 \text{ Nb} \text{F}_5$ 185.82 190 375.82

In an apparatus similar to that as described for the preparation of  $SF_6$  (p. 169), Nb is allowed to react with  $F_2$  at  $300^{\circ}C$ .

The product is stored in sealed copper or iron containers.

PROPERTIES:

Colorless, highly refractive crystals; very hygroscopic, deliquesce on exposure to air. Soluble in water and alcohol with hydrolysis; sparingly soluble in  $CS_2$  and chloroform; hydrolyzes in alkali hydroxide solutions. Concentrated  $H_2SO_4$  dissolves  $NbF_5$ somewhat more readily than  $TaF_5$ .

M.p. 78.9°C, b.p. 233.3°C; d 3.293.

**REFERENCES:** 

- I. O. Ruff and E. Schiller. Z. anorg. allg. Chem. <u>72</u>, 329 (1911);
   O. Ruff and J. Zedner. Ber. dtsch. chem. Ges. <u>42</u>, 492 (1909).
- II. J. H. Junkins, R. L. Farrar, Jr., E. J. Barber and H. A. Bernhardt. J. Amer. Chem. Soc. 74, 3464 (1952).

254

## Potassium Heptafluoroniobate (V) K<sub>2</sub>NbF<sub>7</sub>

 $Nb_2O_5 + 6 HF + 4 KHF_2 = 2 K_2NbF_7 + 5 H_2O$ 265.82 120 312.44 608.18 90.05

Niobium (V) oxide is dissolved in 40% HF solution in a polyethylene dish on a steam bath. A solution of KHF<sub>2</sub> is added until a permanent precipitate is formed. The mixture is then allowed to cool; the product is recrystallized from dilute hydrofluoric acid and the crystals are pressed between filter papers. They are finally vacuum-dried.

SYNONYM:

Potassium niobium heptafluoride.

PROPERTIES:

Formula weight 304.09. Small, very lustrous needles, recrystallizable from hydrofluoric acid. Solubility in water  $(18^{\circ}C)$ 8 g./100 ml. Monoclinic (pseudorhombic) structure.

**REFERENCE:** 

G. Krüss and L. F. Nilson. Ber. dtsch. chem. Ges. 20, 1688 (1887).

## Tantalum (V) Fluoride

TaF₅

 $TaCl_5 + 5 HF = TaF_5 + 5 HCl$ 385.17 100 275.83 182.30

Tantalum (V) chloride (30 g.) is added to 50-60 g. of anhydrous HF contained in a reactor similar to that described for the preparation of TiF<sub>4</sub> (p. 250). The reflux condenser is charged with freezing mixture and the reaction mixture is boiled for several hours until the evolution of HCl ceases. The excess HF is then distilled off through a descending condenser. The reflux condenser is then replaced by a distillation head, and the TaF<sub>5</sub> is distilled off into a platinum crucible. It is stored in sealed copper or iron containers. The yield is 65%, based on TaCl<sub>5</sub>.

PROPERTIES:

Colorless, highly refractive prisms which deliquesce when exposed to air. Dissolves in water with effervescence. Fuming

and concentrated nitric acids do not dissolve  $TaF_5$  as well as water. Concentrated  $H_2SO_4$  dissolves only small amounts of  $TaF_5$ . Alkali hydroxide solutions cause a vigorous reaction. Dissolves to some extent in hot  $CS_2$  and  $CCl_4$ . Reacts vigorously with ether. Attacks glass very slowly at room temperature, but rapidly above.

M.p. 96.8°C, b.p. 229.5°C; d (20°C) 4.74.

**REFERENCES:** 

O. Ruff and E. Schiller. Z. anorg. allg. Chem. 72, 329 (1911).
 O. Ruff and J. Zedner. Ber. dtsch. chem. Ges. <u>42</u>, 492 (1909).

## Potassium Heptafluorotantalate (V)

K<sub>2</sub>TaF<sub>7</sub>

 $\begin{array}{c} Ta_2O_5 + 6 \ HF + 4 \ KHF_2 = 2 \ K_2TaF_7 + 5 \ H_2O \\ 441.76 \quad 60 \quad 312.44 \quad 784.14 \quad 90.05 \end{array}$ 

A platinum dish containing  $Ta_2O_5$  is placed on a water bath and the  $Ta_2O_5$  dissolved in 40% hydrofluoric acid solution; a solution of KHF<sub>2</sub> is added to this mixture until a precipitate forms. The mixture is then allowed to cool. The precipitate of  $K_2TaF_7$  can be recrystallized from hydrofluoric acid. It is pressed dry between filter papers and dried at 120°C.

SYNONYM:

Potassium tantalum heptafluoride.

PROPERTIES:

Formula weight 392.07. Lustrous, thin, short needles, easily recrystallized from hydrofluoric acid. Solubility in water  $(15^{\circ}C)$  0.5 g./100 ml. Monoclinic (pseudorhombic) structure.

REFERENCE:

J. J. Berzelius. Pogg. Ann. 4, 6 (1825).

## Chromium (II) Fluoride

CrF<sub>2</sub>

 $\operatorname{CrCl}_{2} + 2 \operatorname{HF} = \operatorname{CrF}_{2} + 2 \operatorname{HCl}_{122.92}$ 40 90.01 72.92

An apparatus similar to that described for the preparation of  $CoF_2$  (p. 267) is used to pass anhydrous HF over anhydrous  $CrCl_2$ 

until evolution of HCl ceases. The reaction proceeds even at room temperature. The mixture is finally heated in a stream of HF to  $100-200^{\circ}$ C; the excess HF is driven off with a stream of dry nitrogen, in which the product is allowed to cool.

PROPERTIES:

Dark green, crystalline substance with an opalescent luster. Slightly soluble in water, insoluble in alcohol. Not attacked by hot dilute sulfuric or nitric acids. Soluble in boiling hydrochloric acid. Converts to  $Cr_2O_3$  on heating in air.

M.p. 1100°C, b.p. >1200°C; d 4.11. Monoclinic crystals.

**REFERENCE:** 

C. Poulenc. Comptes Rendus Hebd. Séances Acad. Sci. <u>116</u>, 254 (1893).

## Chromium (III) Fluoride

CrF<sub>3</sub>

 $CrCl_3 + 3 HF = CrF_3 + 3 HCl$ 158.38 60 109.01 109.38

In a procedure similar to that described for the preparation of  $CoF_2$  (p. 267),  $CrCl_3$  is heated in a stream of HF until the evolution of HCl ceases. The temperature must be raised to 600°C. The excess HF is then displaced with a dry stream of nitrogen, in which the product is allowed to cool. The product can be melted in an HF stream in a platinum tube at 1200°C and partly distilled off. This treatment yields a crystalline product.

The reaction of chromium hydroxide with hydrofluoric acid yields the trihydrate, not the anhydrous material.

PROPERTIES:

Greenish needles, insoluble in water and alcohol. M.p. >1000°C, b.p. >1100°C; d 3.8.

REFERENCES:

C. Poulenc. Comptes Rendus Hebd. Séances Acad. Sci. <u>116</u>, 254 (1893); Ann. Chim. Phys. (7) 2, 62 (1894).

## Chromium (IV) Fluoride CrF4

 $2 \operatorname{CrCl}_{3} + 4 \operatorname{F}_{2} = 2 \operatorname{CrF}_{4} + 3 \operatorname{Cl}_{2}$ 316.75 152 256.02 212.73

 $\mathbf{or}$ 

 $\begin{array}{c} Cr + 2 F_2 = CrF_4 \\ 58.01 & 76 & 128.01 \end{array}$ 

An apparatus similar to that described for the preparation of  $SF_6$  (p. 169) is used for the fluorination of pulverized electrolytic Cr or  $CrCl_3$  contained in a fluorspar or small alumina vessel. The fluorination temperature is  $350-500^{\circ}C$ . Some  $CrF_4$ , as well as most of the  $CrF_5$ , migrates into the receiver. The main portion of the  $CrF_4$  is deposited in the reaction tube beyond the boat in varnish-like, glittering brown beads. When the fluorination is complete, the apparatus is flushed out with N<sub>2</sub> or  $CO_2$  and the  $CrF_4$  is immediately sealed off in glass ampoules.

PROPERTIES:

Formula weight 128.01. Brown, amorphous, hygroscopic solid, intensely blue vapor; soluble in water (with hydrolysis). M.p. about 200°C, b.p. about 400°C; d 2.89.

REFERENCE:

H. von Wartenberg. Z. anorg. allg. Chem. 247, 136 (1941).

## **Chromyl Fluoride**

CrO<sub>2</sub>F<sub>2</sub>

# $\begin{array}{c} CrO_2Cl_2 + F_2 = CrO_2F_2 + Cl_2 \\ 154.92 & 38 & 122.01 & 70.91 \end{array}$

A stream of nitrogen gas at a rate of about 50 ml./min. is passed through a quartz trap in which  $CrO_2Cl_2$  is kept in a glycerol bath at a temperature not higher than 100°C (see Fig. 138). The nitrogen stream, saturated with  $CrO_2Cl_2$ , is combined in an iron tee joint with a fluorine stream, flowing at a rate of about 60-70 ml./min. The gas mixture flows through a 3-cm.-diameter nickel reaction tube, which is heated electrically to 200°C. A sintered alumina tube equipped with copper caps (see BiF<sub>5</sub>, p. 202) may be used instead of the nickel tube. The reaction mixture is



Fig. 138. Preparation of chromyl fluoride.

then led into a quartz U tube, the arms of which should be at least 15 mm. apart. The products are condensed here with Dry Iceacetone mixture. To avoid contact with atmospheric moisture, an iron drying tube with freshly dehydrated KF is attached to the apparatus.

When sufficient product (a brown feltlike mass) has condensed in the U tube, the glycerol bath is replaced with an ice bath, and  $N_2$  is passed through the apparatus until the  $F_2$  is completely flushed out. The ice bath under the U tube is then removed, while a nitrogen flow is maintained through it. The brown substance decolorizes and forms a gray-white feltlike mass. The nitrogen flow is shut off and both arms of the U tube are fused at their narrowest points. The  $CrO_2F_2$  is stored in the U tube as the white, stable modification.

PROPERTIES:

Reddish-brown vapor. Two solid modifications exist. One is reddish-brown to black-red, unstable (especially when exposed to visible, ultraviolet or infrared light) and can be stored only by chilling the freshly condensed product in liquid nitrogen with exclusion of light. Rhombic or monoclinic crystals. Vapor pressure  $(0^{\circ}C)$  24 mm.

Subl. t. 30°C, m. p. 31.6°C.

The second, polymeric modification is gray-white, stable and starts to volatilize only at  $200^{\circ}$ C, producing red-brown fumes of CrO<sub>2</sub>F<sub>2</sub>.

REFERENCE:

H. von Wartenberg. Z. anorg. allg. chem. 247, 140 (1941).

## Molybdenum (VI) Fluoride Mof<sub>6</sub>

 $\begin{array}{c} Mo + 3 F_2 = MoF_6 \\ 96.0 \quad 114.0 \quad 210.0 \end{array}$ 

Molybdenum metal is fluorinated in an apparatus similar to that described for the preparation of  $SF_6$  (p. 169) (nickel reaction tube,

quartz freezing trap). The Mo powder is introduced into the reactor in a small sintered alumina or platinum boat. The trap is cooled with liquid nitrogen or, in an emergency, with a Dry Ice-acetone bath. When the apparatus is well flushed with  $F_2$  gas, the nickel reaction tube is carefully heated until the reaction starts. The reactor must be cooled occasionally during the fluorination. The simplest way to accomplish this is by wrapping a wet rag around it. White MoF<sub>6</sub> condenses in the quartz trap together with small amounts of oxyfluorides (MoOF<sub>4</sub> and MoO<sub>2</sub>F<sub>2</sub>), formed from the oxygen contained in the fluorine. After the fluorination is complete, the MoF<sub>6</sub> must be repeatedly redistilled in a quartz apparatus in order to remove these impurities.

The purity can be estimated by a melting point determination. The compound is stored in sealed quartz ampoules.

PROPERTIES:

White crystals; very hygroscopic and reactive; reacts with water with vigorous effervescence; forms a blue-white mist in moist air. M.p. 17.5°C, b.p. 35.0°C; d (lig.) (20°C) 2.543.

REFERENCE:

O. Ruff and E. Ascher. Z. anorg. allg. chem. 196, 418 (1931).

## Tungsten (VI) Fluoride

WF<sub>6</sub>

 $W + 3 F_2 = WF_6$ 184.0 114.0 298.0

Tungsten powder contained in a small sintered alumina vessel is burned in a fluorine stream in an apparatus similar to that described for the preparation of  $SF_6$  (p. 169). The compound is purified by repeated distillation.

In addition to determination of the melting point, the molecular weight is determined by vapor pressure measurement in a quartz flask; this is a suitable index for checking the purity.

The product may be stored in glass or quartz ampoules.

**PROPERTIES:** 

Colorless gas, faintly yellow liquid, white solid; very hygro-scopic.

M.p. 2.3°C, b.p. 17.5°C; d (liq.) (15°C) 3.441. Rhombic crystals.

REFERENCES:

- O. Ruff and E. Ascher. Z. anorg. allg. Chem. 196, 413 (1931).
- P. Henkel and W. Klemm, Z. anorg. allg. Chem. 222, 68 (1935).

## Uranium (IV) Fluoride

#### UF4

$UO_3 +$	$2 \operatorname{CCl}_2 F_2 =$	= UF <sub>4</sub> +	$Cl_2 +$	COCl <sub>2</sub>	+ CO2
286.07	$2 \cdot 120.93$	314.07	70.91	98.92	44.01

Dichlorodifluoromethane (Freon 12) is passed through a Hg pressure release valve, a bubble counter and a  $P_2O_5$  tube into a glass or quartz reaction tube (diameter 2.5 cm., length 40 cm.) (see Fig. 139). The reaction tube is inserted into a short electric



Fig. 139. Preparation of uranium (IV) fluoride.

furnace which can be heated to a temperature of  $400^{\circ}$ C. Powdered UO<sub>3</sub> is placed in the reaction tube between glass-wool plugs. The escaping gases are led to the hood.

At the beginning, dry oxygen is passed through the apparatus for one hour, while the furnace is heated to  $400^{\circ}$ C. The oxygen flow is then replaced with CF<sub>2</sub>Cl<sub>2</sub>, which is introduced at a rate of one liter per hour. The reaction starts as soon as the temperature reaches  $400^{\circ}$ C. The progress of the reaction can be followed as the color of the product changes to green.

On completion of the reaction, the product is cooled in a stream of  $CF_2Cl_2$ ; very pure  $UF_4$  is obtained. The yield is almost quantitative.

PROPERTIES:

Green powder, thermally stable up to 1100°C. Converted to  $U_3O_8$  on heating in air.

M.p. >1100°C.

**REFERENCE:** 

H. S. Booth, W. Krasny-Ergen and R. E. Heath. J. Amer. Chem. Soc. 68, 1969 (1946).

## Uranium (VI) Fluoride

UF.

 $\begin{array}{r} U_{3}O_{8} + 3 F_{2} = 3 UF_{6} + 4 O_{2} \\ 842.42 & 542 & 1056.42 & 128 \end{array}$ 

Dried, powdered  $U_3O_8$ , contained in a small nickel boat, is reacted with  $F_2$  gas in an apparatus similar to that described for the preparation of SF<sub>6</sub> (p. 169). The temperature must be carefully maintained above 600°C, since otherwise the oxyfluoride UOF<sub>4</sub> is formed.

The product is collected in a quartz trap. It is then mixed with NaF (to retain the traces of the HF). The  $UF_{e}$  is then repeatedly vacuum-sublimed in a quartz apparatus.

Can be stored in silica ampoules. Larger quantities are preferably stored in an iron container provided with a needle valve.

PROPERTIES:

Formula weight 352.14. White crystals when pure, yellowish when less pure; smokes in air and is hydrolyzed vigorously by water. Does not attack glass if pure.

M.p. 69.5°C (under pressure), subl. t. 56.2°C; d (64.052°C, triple point) 3.63, d (solid) 4.87. Monoclinic crystals.

**REFERENCES:** 

W. Kwasnik. Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review) 23, 18; German Patent Application J 772863.

## Manganese (II) Fluoride MnF<sub>2</sub>

$$\begin{array}{c} MnCO_3 + 2 HF = MnF_2 + CO_2 + H_2O\\ 114.94 & 40 & 92.93 & 44 & 18 \end{array}$$

Manganese carbonate is added to an excess of 40% hydrofluoric acid solution contained in a platinum or lead dish. The pale-red solution of MnF<sub>2</sub> is then decanted and dried at  $110^{\circ}$ C.

PROPERTIES:

Rose-colored, square prisms. Solubility in water 1.06 g./100 ml. Soluble in dilute hydrofluoric acid, readily soluble in concentrated hydrochloric and nitric acids.

M.p. 856°C; d 3.98. Tetragonal (rutile) crystal structure.

**REFERENCE:** 

H. Moissan and Venturi. Comptes Rendus Hebd. Séances Acad. Sci. <u>130</u> b, 1158 (1900).

#### Manganese (III) Fluoride

#### MnF<sub>3</sub>

 $\begin{array}{rrrr} 2 \, \mathrm{MnI_2} + 13 \, \mathrm{F_2} = 2 \, \mathrm{MnF_3} + 4 \, \mathrm{IF_5} \\ 617.54 & 494 & 223.86 & 887.68 \end{array}$ 

Freshly fused and powdered  $MnI_2$ , contained in a small sintered alumina or Pt boat, is fluorinated with  $F_2$  in an apparatus similar to that described for the preparation of  $SF_6$  (p. 169). The reaction is exothermic and  $IF_5$  is evolved. Heat is then applied until 250°C is reached; the product is left to cool in a  $F_2$  stream. The  $F_2$ is finally displaced by dry  $N_2$  and the product is immediately placed in ampoules.

Anhydrous  $MnF_2$  can also be converted to  $MnF_3$  at 250°C, using the same procedure.

R. Hoppe recommends fluorination of  $(NH_4)_2MnF_8$  with elemental  $F_2$  to prepare  $MnF_3$ . This avoids the necessity of observing all the precautions usually required with a hygroscopic starting material. Moreover, the fluorination is more complete, since the molar volume of the starting material is greater than that of the end product.

The product may be stored in sealed glass ampoules.

PROPERTIES:

Formula weight 111.93. Wine-colored; thermally stable to  $600^{\circ}$ C; hydrolyzed by water; d 3.54.
**REFERENCES:** 

- H. Moissan. Comptes Rendus. Hebd. Séances Acad. Sci. 130 c, 622 (1900).
- H. von Wartenberg. Z. anorg. allg. Chem. 244, 346 (1940).
- R. Hoppe. Unpublished private communication.

# Potassium Hexafluoromanganate (IV)

#### K<sub>2</sub>MnF<sub>6</sub>

 $\begin{array}{c} MnCl_2 + 2 KCl + 3 F_2 = K_2 MnF_6 + 2 Cl_2 \\ 125.84 & 149.12 & 114 & 247.16 & 141.8 \end{array}$ 

A mixture consisting of two moles of KCl and one mole of  $MnCl_2$  is heated to 375-400°C in a stream of  $F_2$ , using an apparatus similar to that described for the preparation of  $TlF_3$  (p. 231). After cooling, the excess fluorine is driven off with a stream of dry nitrogen.

SYNONYM:

Potassium manganese hexafluoride.

PROPERTIES:

Gold-yellow, transparent platelets. Turns red-brown when heated but resumes its original color on cooling. Decomposed by water, precipitating hydrated MnO<sub>2</sub>. Hexagonal crystals.

REFERENCE:

E. Huss and W. Klemm. Z. anorg. allg. Chem. 262, 25 (1950).

# Rhenium (VI) Fluoride

ReF.

 $Re + 3 F_2 = ReF_6$ 186.3 114 300.3

A small fluorspar boat, containing Re powder, is placed in a fluorspar tube and oxygen-free fluorine gas is passed through. Since removal of  $O_2$  from the crude fluorine gas is carried out at reduced pressure, the fluorination of Re must also be done under reduced pressure. Because of this the fluorspar tube must be encased in a nickel or copper tube. Fluorine gas, kept condensed in a quartz trap immersed in liquid nitrogen (see Fig. 140), is led at a pressure of 20-35 mm. through a spark gap (about 5000 v., 0.012 amp.), also immersed in liquid nitrogen. The oxygen in the crude fluorine gas is thus converted to  $O_2F_2$  and frozen out. The purified fluorine gas sweeps over the rhenium powder, heated to  $125^{\circ}$ C by an electric furnace. The gaseous reaction products pass through two quartz traps, cooled with liquid nitrogen, where they are condensed. The two condensation traps are connected to an additional quartz trap, which prevents access of atmospheric moisture. A stopcock closes the system or serves as a connection to an aspirator.



Fig. 140. Preparation of rhenium (VI) fluoride.

The ground joints of the apparatus are not greased but are sealed externally with picein. The water film is removed from the walls of the entire apparatus by heating in a stream of nitrogen before the beginning of the run.

As soon as the Repowder comes in contact with the fluorine gas, a white, blue and violet fog appears in the condensation traps. This is followed at once by the almost colorless  $ReF_6$ .

After the reaction, the apparatus is flushed with dry nitrogen to remove the excess fluorine. The  $\text{ReF}_6$  is then resublimed in quartz apparatus under vacuum. Because  $\text{ReF}_6$  reacts readily with quartz, this treatment is performed only once. Fractional distillation is inapplicable in this case due to the formation of  $\text{ReOF}_4$ .

The product is stored in quartz ampoules placed in liquid nitrogen.

#### PROPERTIES:

Pale yellow, featherlike, crystalline powder, extremely hygroscopic; fumes in air with formation of blue smoke, which later turns dark violet. Nitric acid dissolves  $\text{ReF}_6$  with the formation of white smoke. Very corrosive to glass. Attacks quartz slightly. Copper is stable to it up to 150°C. Instantly blackened by concentrated H<sub>2</sub>SO<sub>4</sub>, benzene, acetic acid, paraffin oil, alcohol, ether and acetone.

M.p. 18.8°C, b.p. 47.6°C; d (liq.) (19°C) 3.616.

**REFERENCES:** 

- O. Ruff and W. Kwasnik. Z. anorg. allg. Chem. 209, 113 (1932).
- O. Ruff and W. Kwasnik. Z. anorg. allg. Chem. <u>219</u>, 65 (1934).

# Iron (II) Fluoride

#### FeF<sub>2</sub>

 $FeCl_2 + 2 HF = FeF_2 + 2 HCl$ 126.76 40 93.84 72.92

Iron (II) chloride is treated with anhydrous HF in an apparatus similar to that described for the preparation of  $CoF_2$  (Fig. 141). The reaction proceeds even at room temperature, yielding an amorphous product. To obtain crystals, the product must be heated to  $1000^{\circ}C$ .

PROPERTIES:

White powder, sparingly soluble in water, insoluble in alcohol, ether and benzene.

M.p. >1100°C, subl. t. about 1100°C; d 4.09. Tetragonal (rutile) structure.

REFERENCES:

- C. Poulenc. Comptes Rendus Hebd. Séances Acad. Sci. <u>115</u>, 942 (1892).
- C. Poulenc. Ann. Chim. Phys. (7) 2, 53 (1894).

# Iron (III) Fluoride

# FeF<sub>3</sub>

 $FeCl_3 + 3 HF = FeF_3 + 3 HCl$ 162.22 60 112.84 109.38

Anhydrous  $\text{FeCl}_3$  is allowed to react with anhydrous HF in an apparatus similar to that described for the preparation of  $\text{CoF}_2$  (Fig. 141) until HCl evolution ceases. The reaction proceeds even at room temperature and yields amorphous  $\text{FeF}_3$ . In order to obtain crystals, the product must be heated to  $1000^{\circ}\text{C}$ .

PROPERTIES:

Greenish powder, very slightly soluble in water (at  $25^{\circ}$ C, 0.091 g./100 ml.), readily soluble in dilute hydrofluoric acid, insoluble in alcohol, ether and benzene.

Subl. t. >1000°C; d 3.87. Hexagonal crystals.

REFERENCES:

C. Poulenc. Comptes Rendus Hebd. Séances Acad. Sci. <u>115</u>, 944 (1892). C. Poulenc. Ann. Chim. Phys. (7) <u>2</u>, 57 (1894).

### Cobalt (II) Fluoride

#### CoF<sub>2</sub>

 $CoCl_2 + 2 HF = CoF_2 + 2 HCl$ 122.89 40.02 96.97 72.94

Crystalline cobalt (I) chloride  $(CoCl_2 \cdot 2 H_2O)$  is completely dehydrated at 200°C in a glass tube through which a HCl stream is passed. The progress of the dehydration can be easily followed, since the color changes from pink to blue.



Fig. 141. Preparation of cobalt (II) fluoride.

After this, anhydrous HF at  $300^{\circ}$ C is passed over the CoCl<sub>2</sub>, contained in a small fluorspar boat placed inside an iron tube, until evolution of HCl is no longer detectable at the end of the tube. The apparatus is then flushed with dry nitrogen to remove excess HF.

This apparatus is suitable for all fluorinations with anhydrous HF in which the product is a nonvolatile solid fluoride ( $CrF_2$ ,  $CrF_3$ ,  $VF_3$ ,  $FeF_2$ ,  $FeF_3$ ).

PROPERTIES:

Reddish-pink powder, sparingly soluble in water; dissolves readily in mineral acids on heating.

M.p. 1100-1200°C; d 4.43. Tetragonal (rutile) structure.

**REFERENCES:** 

O. Ruff and E. Ascher. Z. anorg. allg. Chem. <u>183</u>, 193 (1929). W. B. Burford. Ind. Eng. Chem. 39, 321 (1947).

# Cobalt (III) Fluoride

CoF<sub>3</sub>

I.

 $\begin{array}{r} 2\,{\rm CoF_2}\,+\,{\rm F_2}=\,2\,{\rm CoF_3}\\ 193.88 & 38.0 & 231.88 \end{array}$ 

Cobalt (II) fluoride (see above) is treated with  $F_2$  gas in an apparatus similar to that described for the preparation of  $TlF_3$  (p. 231). At first the conversion proceeds rather slowly, but it becomes vigorous when the reaction tube is heated to 75°C. Due to the heat of reaction, the temperature rises to 200°C. The product is cooled in a stream of fluorine, and the excess of the latter is then flushed out with dry N<sub>2</sub>. The yield is 91%, based on CoF<sub>2</sub>.

II.	$2 \operatorname{CoCl}_2$	$+ 3 F_2 =$	= 2 CoF <sub>3</sub>	$+ 2 \operatorname{Cl}_2$
	259.71	114	231.88	141.8
or:	Co <sub>2</sub> O <sub>3</sub> 165.88	$+ 3 F_2 = 114$	2 CoF <sub>3</sub> + 231.88	$\frac{1^{1/2}O_{2}}{48.0}$

Anhydrous  $CoCl_2$  or  $Co_2O_3$  is treated with  $F_2$  gas in an apparatus similar to that described for the preparation of  $TlF_3$  (p. 231). The temperature of the reaction tube is raised gradually to 300°C, starting from room temperature; this temperature is maintained until  $F_2$  gas can be detected at the exit. The excess  $F_2$  in the apparatus is then displaced with dry  $N_2$ .

The product is stored in hermetically sealed glass, quartz or metal containers.

Used to fluorinate organic compounds.

PROPERTIES:

Formula weight 115.94. Light-brown powder, becomes dark brown on exposure to moist air; volatilizes in a stream of  $F_2$  at  $600-700^{\circ}C$ ; decomposes extensively at lower temperatures into  $CoF_2 + F_2$ . Reacts with water with evolution of  $O_2$ .

d 3.88. Hexagonal crystals.

REFERENCES:

I. E. A. Belmore, W. M. Ewalt and B. H. Wojcik. Ind. Eng. Chem. <u>39</u>, 341 (1947).

II. O. Ruff and E. Ascher. Z. anorg. allg. Chem. <u>183</u>, 193 (1929);
 E. T. McBee et al. Ind. Eng. Chem. <u>39</u>, 310 (1947).

### Nickel (II) Fluoride

#### NiF<sub>2</sub>

 $\begin{array}{l} NiCl_2 + F_2 = NiF_2 + Cl_2 \\ 126.6 & 38 & 96.69 & 70.92 \end{array}$ 

Anhydrous NiCl<sub>2</sub>, contained in a small nickel boat, is fluorinated at 150°C in an apparatus similar to that described for the preparation of TF<sub>3</sub> (p. 231). The reaction product remaining in the boat (its composition is approximately NiF<sub>2.5</sub>) is then heated in a stream of N<sub>2</sub> or CO<sub>2</sub>, yielding NiF<sub>2</sub> and splitting off F<sub>2</sub>.

PROPERTIES:

Yellowish-green powder, sparingly soluble in water, insoluble in alcohol and ether; sublimes in a stream of HF above 1000°C. d 4.63. Tetragonal (rutile) structure.

**REFERENCE:** 

P. Henkel and W. Klemm. Z. anorg. allg. Chem. 222, 74 (1935).

# Potassium Hexafluoronickelate (IV)

#### K<sub>2</sub>NiF<sub>6</sub>

A mixture consisting of two moles of KCl and one mole of NiCl<sub>2</sub> is heated for three hours in a stream of fluorine at  $275^{\circ}C$  in an apparatus similar to that described for the preparation of TlF<sub>3</sub> (p.231). After cooling, the excess fluorine is expelled with dry N<sub>2</sub>.

The product is stored in glass ampoules sealed in vacuum or in glass bottles sealed under nitrogen.

This method of preparation is quite versatile and may be applied, for example, to the production of  $K_2MnF_6$ ,  $K_2CrF_6$ ,  $K_2FeF_6$ ,  $K_3CoF_7$ ,  $K_2VF_6$  and  $K_3CuF_6$ .

PROPERTIES:

Lustrous red salt, reacts with water, with evolution of gas (OF<sub>2</sub>?) and formation of a black precipitate; reduced by H<sub>2</sub> at 200°C. d 3.03. Has a K<sub>2</sub>PtCl<sub>2</sub>-type structure.

REFERENCE:

W. Klemm and E. Huss. Z. anorg. allg. Chem. 258, 221 (1949).

# Iridium (VI) Fluoride IrFs

 $Ir + 3 F_2 = IrF_6$ 193.1 114 307.1

Iridium, contained in a small fluorspar boat placed in an electrically heated fluorspar tube, is fluorinated at 240°C (Fig. 142). A nickel or platinum tube may be employed instead of the fluorspar reactor. The fluorine gas first passes through a quartz trap



Fig. 142. Preparation of iridium (VI) fluoride.

A, maintained at -170 to  $-196^{\circ}$ C in order to freeze out the HF. From there it flows to the reactor furnace. The product gases flow through two silica U tubes or gas traps (I and II). The temperatures of I and II are  $-78^{\circ}$ C and  $-196^{\circ}$ C, respectively. Terminal trap B (maintained at  $-196^{\circ}$ C), serves to prevent access of atmospheric moisture.

The fluorspar tube is connected to the quartz sections of the apparatus with ground joints, which are externally sealed with an asbestos-waterglass mixture.

The apparatus is first flushed with dry nitrogen, while the quartz sections are heated to remove traces of surface moisture. The flow of fluorine gas is then started and the traps are cooled. Yellow vapors of  $IrF_{\rm s}$  appear as soon as the fluorine reaches the Ir. These

collect in the traps. After the fluorination, the excess  $F_2$  is purged with  $N_2$ . The  $IrF_6$  is then purified by fractional distillation in a quartz vacuum apparatus without stopcocks, and it is finally distilled into quartz ampoules, which are immediately sealed off.

PROPERTIES:

Bright yellow, lustrous lamellae and small needles, which above  $-15^{\circ}$ C become intensely gold-yellow and glassy. Very hygroscopic; attacks glass. Corrodes Pt at temperatures above 400°C. Reduced by halogens to IrF<sub>4</sub> at room temperature.

M.p. 44.4°C, b.p. 53°C; d (solid) (-190°C) about 6. Tetragonal crystals.

REFERENCE:

O. Ruff and J. Fischer. Z. anorg. allg. Chem. 179, 166 (1929).

# SECTION 5

Chlorine, Bromine, Iodine

## M. SCHMEISSER

# Chlorine

#### $\mathbf{Cl}_{\mathbf{2}}$

Commercially available liquid chlorine, which is obtained by electrolysis of alkali, is not sufficiently pure and must therefore be purified by method I.

On the other hand, a gas that is already largely free of such impurities as  $O_2$  and chlorine oxides is produced by the reaction of hydrated manganese dioxide with pure hydrochloric acid. For this preparation, see method II below.

I. Chlorine from a steel cylinder is passed consecutively through two wash bottles or columns containing concentrated  $H_2SO_4$ , a tube or column containing CaO (to remove any HCl that might be present), a tube containing  $P_2O_5$ , and finally into a container placed in a Dry Ice-acetone bath, where it is condensed and liquefied. The liquefied  $Cl_2$  is repeatedly vaporized and condensed while noncondensable gases ( $O_2$ ) are continuously removed with a pump. Finally, the liquid  $Cl_2$  is fractionated in high vacuum and passed into receivers cooled with liquid nitrogen. (For the apparatus see, for example, Part I, p. 66 ff.) Only the middle fraction is used for further work.

II. 
$$MnO_2 \cdot x H_2O * + 4 HCl = MnCl_2 + (x+2)H_2O + Cl_2 - 100 145.88$$
 70.91

 $x \sim 0.8$  for a product of about 86% purity.

Concentrated, air-free hydrochloric acid (d 1.16) is added dropwise to precipitated hydrated manganese dioxide (e.g., the 86% pure commercially obtainable material) in a flask equipped with a dropping funnel and a gas outlet tube. The gas formation may be regulated by moderate heating.

The chlorine thus formed is passed through water (to remove entrained HCl) and  $H_2SO_4$  (carried out as in method I, that is,

 $\rm H_2SO_4,$  a tube containing CaO, a tube with  $\rm P_2O_5)$  and liquefied in a receiver cooled with a Dry Ice-acetone bath. Subsequent purification is as in method I.

Other preparative methods:

III. Electrolysis of an NaCl solution saturated with HCl in the electrolytic cell described by Bodenstein and Pohl. The oxygen content of the  $Cl_2$  produced in this manner is 0.01%.

Extremely pure Cl<sub>2</sub> can be produced in small quantities by the following methods:

IV. Heating  ${\rm AuCl}_3$  (prepared from finely divided Au and dry  ${\rm Cl}_2$ ) at 250°C in vacuum.

V. Sublimation-crystallization procedure carried out in high vacuum. (In this process, the purity of the  $Cl_2$  product is checked by measuring the rate of formation of phosgene from CO and  $Cl_2$ . This reaction is retarded by the slightest impurities.)

Klemenc considers the most effective means of removing the last traces of  $O_2$  from  $Cl_2$  to be the bubbling of very pure  $H_2$  through liquid  $Cl_2$  at  $-78^{\circ}C$  for 24 hours.

#### PROPERTIES:

Yellow-green, pungent gas. M. p.  $-101.0^{\circ}$ C, b. p.  $-34.0^{\circ}$ C. Heat of fusion 1531 cal./mole; heat of vaporization 4878 cal./mole. Triple point pressure 10.4 mm., crit. t. 143.5°C, crit. p. 76.1 atm. d,(liq.) ( $-34^{\circ}$ C) 1.557. Solubility in water: 1 vol. of water dissolves 4.6 vol. of Cl<sub>2</sub> at 0°C, 2.15 vol. at 20°C, 1.22 vol. at 50°C, 0.39 vol. at 90°C.

Chlorine attacks rubber, cork, stopcock grease and Hg but can be stored in glass containers over concentrated  $H_2SO_4$  or as a liquid in steel cylinders. The vigorous reaction of chlorine with many commonly used metals occurs only at elevated temperatures; the reaction with steel, for example, starts above 250°C [G. Heinemann, F. G. Garrison and P. A. Haber, Ind. Eng. Chem., Ind. Ed. 38, 497 (1946)].

REFERENCES:

- I. A. Klemenc. Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], 2nd ed., Vienna, 1948, p. 153.
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   W. F. Giauque and T. M. Powell. J. Amer. Chem. Soc. <u>61</u>, 1970 (1939).
- III. M. Bodenstein. Z. Elektrochem. 22, 204 (1916).
- IV. A. Coehn and G. Jung. Z. phys. Chem. <u>110</u>, 705 (1924).
   H. von Wartenberg and F. A. Henglein. Ber. dtsch. chem. Ges. 55, 1003 (1922).

V. P. M. Fye and J. J. Beaver. J. Amer. Chem. Soc. <u>63</u>, 1268 (1941).

# **Chlorine Hydrate**

 $Cl_2 \cdot 6 H_2O$ 

 $\begin{array}{c} Cl_2 \ + \ 6 \ H_2O \ = \ Cl_2 \ \cdot \ 6 \ H_2O \\ 70.9 \ 108 \ 178.9 \end{array}$ 

I. Chlorine is dissolved in water at  $0^{\circ}$ C, forming a thin slurry which is then filtered through a glass filter funnel surrounded by a jacket cooled with ice water. The crystals, which are thus largely freed from water, are sealed into a glass tube and heated to 30 to  $40^{\circ}$ C. Decomposition into liquid Cl<sub>2</sub> (under its own pressure) and Cl<sub>2</sub>-saturated water results. The sealed tube is allowed to cool from 40 to  $0^{\circ}$ C in a large water bath for two days. Thus, the mixture components recombine and form larger crystals.

II. Better-formed crystals can be prepared in the following way:

Chlorine hydrate, prepared as above, is placed in one arm of a thick-wall U tube and the tube is sealed off. The hydrate is decomposed by heating and the chlorine formed is condensed by immersing the other arm of the U tube in a refrigerant. Then the refrigerant is removed while the other side of the tube, which contains water saturated with  $Cl_2$ , is immersed in a vessel full of cold water. After some time large, very glittering, pale-yellow crystals are formed in this arm.

#### PROPERTIES:

Yellow crystals. Decomposition temperature at 1 atm. +9.6°C; critical decomposition point 28.7°C, 6 atm.; dissociation pressure (at 0°C) 252 mm.; d. (calc.) 1.29. Cubic crystals, with the theoretical composition of  $Cl_2 \cdot 5^{3}/_4$  H<sub>2</sub>O.

**REFERENCES:** 

- I. E. Biewend. J. prakt. Chem. 15, 440 (1838).
  - H. W. B. Roozeboom. Rec. Trav. Chim. Pays-Bas 3, 59 (1884); 4, 65 (1885).
- II. A. Ditte. Compt. Rend. Hebd. Séances Acad. Sci. <u>95</u>, 1283 (1882).
  P. Villard. Ann. Chim. (7) <u>11</u>, 292 (1897).
  Schröder. Die Chemie der Gashydrate [Chemistry of Hydrates of Gases], Stuttgart, 1926.
  M. von Stackelberg. Naturwiss. 36, 327, 359 (1949).

M. von Stackelberg and R. H. Müller. Z. Elektrochem. <u>58</u>, 25 (1954).

### Bromine

#### Br<sub>2</sub>

Even the purest commercial bromine contains approximately 0.05% Cl as well as traces of I, and must therefore be purified for special uses.

I. In order to remove most of the still present chlorine, bromine may be stored with pulverized KBr for a considerable time and then distilled off in high vacuum into a receiver cooled with a Dry Ice-ether mixture.

II. Very pure bromine may be prepared, according to Hönigschmid and Baxter, in the following manner: A concentrated solution of CaBr, or KBr is placed in a round-bottom flask connected with ground joints to a bromine-containing dropping funnel and to an exit tube, bent at right angles. The tube passes through a condenser into a receptacle containing ice-cold, very pure water. [The very pure CaBr, starting material is prepared by dropwise addition of bromine (which has been subjected to the purification described above) to ammoniacal calcium hydroxide. The calcium hydroxide is prepared from the very purest, halogen-free line.] Bromine is added from the funnel to the flask and is then distilled off from the solution. As the Br<sub>2</sub> distills off, more bromine is added below the surface of the CaBr<sub>2</sub> (or KBr) solution from the dropping funnel. The distilled bromine is reduced to KBr by dropwise addition to a hot solution of recrystallized, halogen-free potassium oxalate. The KBr solution is evaporated. During evaporation, small quantities of Br<sub>2</sub> are liberated frequently by addition of acidified KMnO<sub>4</sub> solution, which through evaporation also removes any I2 that may be present. According to Baxter, small quantities of absolutely pure Br, from a previous run may be added to achieve the same result. In order to decompose traces of organic materials, the KBr that crystallizes out is fused in a Pt crucible. It can then be considered completely free from Cl and I.

Bromine is now liberated by treatment of the KBr with very pure  $K_2Cr_2O_7$  and very pure  $H_2SO_4$  (the latter is obtained by distillation over  $K_2Cr_2O_7$ , discarding the forerun). However, the reaction with  $K_2Cr_2O_7$  is not complete, since only about 75% of the needed  $K_2Cr_2O_7$  enters into the reaction. Thus, the remaining  $Br_2$  must be distilled again from the KBr solution formed. The product  $Br_2$  is washed with water to remove HBr, separated from the entrained  $H_2O$ , and then dried over very pure CaO and CaBr<sub>2</sub> or over  $P_2O_5$ . Finally, it is freed of these substances by distillation in vacuum. PROPERTIES:

Formula weight 159.84. Reddish-brown, pungent liquid. M.p.  $-7.3^{\circ}$ C, b.p. 58.8°C; d (0°C) 3.19. Solubility in water (20°C) 3.53 g. of Br<sub>2</sub> per 100 g. of H<sub>2</sub>O.

**REFERENCES:** 

- I. W. A. Noyes, Jr. J. Amer. Chem. Soc. 45, 1194 (1923).
- II. O. Hönigschmid and E. Zintl. Liebigs Ann. Chem. 433, 216 (1923).

G. P. Baxter, C. J. Moore and A. C. Boylston. J. Amer. Chem. Soc. <u>34</u>, 260 (1912).

# **Bromine Hydrate**

# $\mathbf{Br_2} \cdot \mathbf{8} \mathbf{H_2O}$

 $\begin{array}{rrr} Br_2 \,+\,8\,H_2O \,=\,Br_2\,\cdot\,8\,H_2O \\ 159.8 & 144 & 303.8 \end{array}$ 

A 4% (by weight) solution of  $Br_2$  in water (saturated solution at 0°C) is cooled to 0°C. This causes a small quantity of bromine hydrate (about 4% of the  $Br_2$ -H<sub>2</sub>O mixture) to separate out. Usually, however, the solution must be either seeded with some bromine hydrate or cooled for a short time to -5°C, after which the temperature is restored to 0°C. The precipitate is filtered on a glass filter funnel surrounded by a jacket containing ice water.

In order to form larger crystals, the product hydrate is sealed into a tube together with a large excess of 4% bromine water and kept for about four weeks on ice. The tube is warmed to  $5-6^{\circ}$ C once each day.

PROPERTIES:

Light-red crystals, which must be stored in a sealed tube at temperatures below  $6.2^{\circ}$ C. Critical decomposition point:  $6.2^{\circ}$ C, 93 mm.; dissociation pressure (0°C), 45 mm.; d (solid) (0°C) 1.49.

The composition is somewhat uncertain. Cubic crystals have the theoretical composition  $Br_2 \cdot 7^2/_3 H_0 O_1$ .

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- H. Giran. Comp. Rend. Hebd. Seances Acad. Sci. 159, 246 (1914).

W. H. Harris, J. Chem. Soc. (London) 1933, 582.

M. von Stackelberg. Naturwiss. <u>36</u>, 327, 359 (1949).

M. von Stackelberg and H. R. Müller. Z. Elektrochem. 58, 25 (1954).

# lodine

#### **I**2

Since even the purest commercial KI to be used for the preparation of specially purified iodine may still contain such impurities as Cl, Br, ICN, alkali sulfate, carbonate and sulfide, as well as traces of organic material, special purification is necessary. I. Preparation of very pure iodine according to Hönigschmid.

A supersaturated solution of C.P. purity KI is mixed with a solution of thrice recrystallized, completely halogen-free  $CuSO_4 \cdot 5H_2O$ . The CuI formed is allowed to settle and the supernatant solution of  $I_2$  in KI is decanted and distilled. The  $I_2$  is steam-distilled. The water is decanted from the condensate and the  $I_2$  is again distilled from the KI solution and finally from pure water. After filtering through a glass frit, the iodine is dried in a desiccator over concentrated  $H_2SO_4$  and finally sublimed in a quartz tube in a stream of nitrogen.

II. If extreme purification is unnecessary, commercial iodine or iodine regenerated from wastes can, according to a method described by Plotnikow, be sublimed, first over KI and then over BaO. It is then stored in ground glass containers placed in a desiccator over  $P_2O_5$ .

#### PROPERTIES:

Formula weight 253.84. Gray-black flakes with a metallic sheen. M.p. 113.7°C, b.p. 184.4°C; d 4.93. Solubility (20°C) 0.29 g./100 ml. of  $H_2O$ .

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- O. Hönigschmid and W. Striebel. Z. phys. Chem. (A) <u>156a</u> (Bodenstein Anniversary Volume), 286 (1931).
- M. Guichard. Ann. Chim. et Phys. (9) 7, 28 (1916).
- W. A. Plotnikow and W. E. Rokotjan. Z. phys. Chem. <u>84</u>, 365 (1913).

#### RECOVERY OF IODINE FROM LABORATORY WASTE SOLUTIONS

In the Arndt method, the oxidation of iodide residues with elementary oxygen, using nitric oxides as carriers, proceeds in accordance with the following reactions:

1.  $HI + HNO_2 = 1/2I_2 + NO + H_2O$ 

2.  $NO + \frac{1}{2}O_2 = NO_2$ 

- 3.  $2 \text{HI} + \text{NO}_2 = I_2 + \text{NO} + H_2\text{O}$
- 4.  $2 \text{ NO}_2 + \frac{1}{2} \text{ O}_2 + \text{H}_2 \text{O} = 2 \text{ HNO}_3$

This procedure is feasible because reaction 4 proceeds slowly in relation to reactions 1-3 so that, as long as iodine is present, no significant loss of nitric oxides occurs.

A large flask is closed off with a rubber stopper. A gas inlet is inserted through the stopper, reaching almost to the bottom; this tube is attached to the inlet tube of an empty wash bottle by a fairly long piece of flexible tubing; the other tube of the wash bottle is attached to a gasometer filled with O<sub>2</sub> from a cylinder. The alkaline solution of iodine residues, after concentration by evaporation, is placed in the flask, which should be no more than half full. It is then acidified with concentrated H<sub>2</sub>SO<sub>4</sub> and the free space of the flask-with the stopper left loose-is filled with oxygen. The gasometer stopcock is then closed and nitrite solution is added to the flask until the free space acquires an intense reddish-brown color. The stopper is then pushed firmly down and the gasometer cock opened. The oxygen begins to flow into the closed flask either immediately or after very slight rotation of the stopcock. The flask is shaken, at first gently and then vigorously and continuously. The rate of oxygen absorption is checked from time to time by interrupting the shaking to determine whether O<sub>2</sub> is still flowing rapidly in and whether the gaseous phase is still red-brown. Should this not be the case, due to the accumulation of inert gases (from the  $N_2$  in the  $O_2$  used or from reduction of a small quantity of the nitric oxide to N<sub>2</sub>O or N<sub>2</sub>), the stopper is raised for a moment. If this does not restore the O2 absorption and the formation of NO2, the stopcock is closed, further nitrite solution is added, and the procedure is continued. If shaking is started or stopped too quickly, some liquid may be driven into the wash bottle due to a temporary rise in pressure. However, the O<sub>2</sub> stream which again starts to flow drives it back into the flask. Completion of the oxidation may be recognized by the cessation of O<sub>2</sub> absorption and by the fact that the gaseous phase becomes colorless. After the black, crystalline iodine has settled, the completeness of the iodine precipitation may be checked by adding a few drops of nitrite solution. The mother liquor (which contains only about 0.5 g. of iodine per liter) is decanted and the iodine precipitate is placed in a round-bottom flask, where it may be combined with iodine prepared in other runs. The  $I_2$  is then steam-distilled from this flask. No condenser is used; instead, the vapor mixture is passed through a large tube (10-15 mm. in diameter) directly into the center of a large two-liter Erlenmeyer flask which is closed with a paraffin-coated cork stopper and immersed in a bath with flowing water. A second hole in the cork stopper contains a vent tube about 0.5 m. long and 1 cm. in diameter. The  $I_2$  separates on the walls as a compact mass. The substance may easily be detached from these surfaces by shaking and cooling. It is crushed with a glass rod and suction-dried while pressing the water out.

The filtered I<sub>2</sub> is given a preliminary and final drying over  $CaCl_2$  or concentrated  $H_2SO_4$  in an ungreased desiccator and then sublimed. The iodine is placed in a spoutless beaker, which is immersed in a hot water bath while a round-bottom flask, filled with cold water, is set on top of the beaker. The flask becomes covered with moisture and some iodine and is replaced with a second flask before the condensed water can drop back. This is repeated as long as moisture is evolved. The end of the water evolution can be recognized by the fact that dry I<sub>2</sub> sticks firmly to the water-cooled glass, whereas moist iodine may easily be washed off the glass with a stream from a wash bottle. The wet beaker is now carefully wiped and carefully heated on an asbestos wire gauze. The round-bottom flask is now cooled on the inside with flowing water. As soon as a 0.5-1 cm. crust of iodine forms, it is scraped off and put in a storage flask. The sublimation is then continued until all the I<sub>2</sub> in the beaker has sublimed.

The method described must be modified in some cases, e.g.: If the iodine residues contain considerable quantities of Fe, the oxidation must be carried out with heating in order to decompose Fe-NO complexes. If Hg or Pb salts are present, the procedure described in Chemiker-Ztg. 47, 16 (1923) is used.

Other preparative methods: A procedure for the recovery of I<sub>2</sub> (and Ag) from AgI residues is given by J. R. Spies, Ind. Eng. Chem., Anal. Ed. 7, 118 (1935); J. R. Spies in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 6.

If iodine is to be recovered from organic iodine compounds, the organic portion is decomposed with a  $KClO_3-Cl_2$  mixture [E. M. Marshall, J. Chem. Ed. 7, 1131 (1930)].

Another procedure, based on reaction with  $Cl_2$ , is described by C. de Witt, J. Chem. Ed. 14, 215 (1937).

In collecting the iodine residues the greatest care should be taken to avoid the presence of any volatile organic solvents in the container. If nonvolatile organic materials such as starches are absent, the steam distillation described above may be omitted. Great care should be taken to avoid contaminating the iodine residues with ammonium salts. Violent explosions may be caused by the formation of nitrogen iodide.

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# Hydrogen Chloride

#### HCl

I. An easily controllable stream of hydrogen chloride gas may be readily obtained by allowing pure, concentrated hydrochloric acid to flow into concentrated  $H_2SO_4$ .

An essential constituent of the apparatus shown in Fig. 143 is the capillary tube. This must be completely filled with hydrochloric acid before the evolution is started, in order to assure the hydrostatic pressure necessary to cause the lighter hydrochloric acid to flow to the bottom of the vessel which contains the heavier  $H_2SO_4$ . Only by allowing the acid to flow in this way is the generation of the gas completely uniform and controllable.

A separatory funnel is filled with approximately 200 ml. of concentrated H<sub>2</sub>SO<sub>4</sub>, and concentrated hydrochloric acid (d, 1, 18)is added from a dropping funnel at such a rate as to give the gas flow desired. When 200 ml. of concentrated hydrochloric acid (i.e., the same volume as the volume of  $H_2SO_4$  used) has been added, gas evolution stops and the dilute sulfuric acid, which now contains very little HCl. is discharged and replaced by fresh H<sub>2</sub> SO<sub>4</sub>. (If more than an equal volume of hydrochloric acid is added, HCl continues to be formed for a while after the stopcock is closed; however, the yield is reduced.) The yield from 200 ml, of concentrated hydrochloric acid is 67.4 g. of HCl.

If a uniform stream of HCl is required for a longer period, the apparatus designed by Seidel (Fig. 144) is recommended. Con-

centrated hydrochloric acid and concentrated sulfuric acid are dropped continuously from tubes a and b into the reaction tube,



Fig. 143. Preparation of hydrogen chloride.

which is about 5 cm. in diameter and contains a 20- to 25-cm. layer c of packing, such as silica or glass beads. The spent liquid mixture automatically drains off below. With an apparatus of these



Fig. 144. Preparation of hydrogen chloride.a, b) Dropping tubes for concentrated HCl and concentrated H<sub>2</sub>SO<sub>4</sub>; c) reaction tube packing (silica or glass beads, diameter 2-5 mm.).

#### PROPERTIES:

dimensions, up to three liters of HCl gas can be produced per minute.

In order to remove moisture that may be present, the product gas is led through a wash bottle containing concentrated  $H_2SO_4$  $(P_2O_5$  must not be used because the gas forms volatile phosphorus compounds with it) and into a receiver chilled with liquid nitrogen. The receiver is then detached from the generator and the gas is fractionally distilled. Only the middle fraction is pure enough for use in further work. (For the apparatus see Part I, p. 66 ff.)

If an especially pure product is not required, the ground glass part of the separatory funnel in Fig. 143 may be replaced by a two-hole rubber stopper. The freezing and fractional distillation of the hydrogen chloride may be omitted in this case.

Formula weight 36.47. Colorless, pungent gas. M.p.  $-112^{\circ}$ C, b.p.  $-85.0^{\circ}$ C; crit. t. 51.3°C, crit. p. 83 atm.; d. (liq.) ( $-85^{\circ}$ C) 1.189. Solubility in water: 1 vol. ( $15^{\circ}$ C) dissolves about 450 vol. of HCl (47% by weight).

Attacks rubber and stopcock grease; glass stopcocks should therefore be lubricated with concentrated  $H_2SO_4$ . The gas can be stored over Hg or over  $H_2SO_4$ .

II. According to Hönigschmid, very pure aqueous solutions of HCl can be obtained by diluting pure laboratory hydrochloric acid to 20% with water, boiling it with small amounts of  $KMnO_4$  to remove bromine and iodine, and then distilling it through a quartz condenser. If the purified hydrochloric acid prepared in this way is needed in more concentrated form, HCl gas is generated from this dilute solution with  $H_2SO_4$  according to the method given under I, and this product gas is then bubbled through purified 20% hydrochloric acid until the latter becomes saturated.

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# Hydrogen Bromide

#### HBr

The method chosen for producing hydrogen bromide depends upon whether it is to be anhydrous or in aqueous solution, as well as on the amount required and the requisite degree of purity of the product.

Methods I and II, which are suitable for the preparation of anhydrous HBr, may also be modified to give aqueous solutions, but the special procedures for obtaining aqueous solutions (V) cannot be modified to give anhydrous HBr. However, regardless of the manner in which they have been prepared, HBr solutions can be dehydrated with  $P_2O_5$  via method III.

While method I (tetralin plus  $Br_2$ ) is very convenient, it should be realized that half of the  $Br_2$  input is lost by reaction with the tetralin. Therefore, method II ( $H_2 + Br_2$ ) is preferred for preparing larger quantities of HBr.

How far the described procedures can be simplified if a highly purified product is not required will be indicated under the respective methods.

I. Preparation of anhydrous HBr from tetralin (tetrahydronaphthalene) and Br<sub>2</sub>:

Bromine is gradually added by drops to a mixture of tetralin and pure iron filings contained in a round-bottom flask equipped with a dropping funnel and a gas outlet tube. (Prior to use, the tetralin is dried over anhydrous  $Na_2SO_4$  and distilled; b.p. of the tetralin is 207°C, vapor pressure at 15°C, 0.3 mm.; C. P. grade  $Br_2$  should be used.) Since initial cooling is necessary, the flask is placed in a water bath, which, as soon as the reaction becomes sluggish, is heated to 30 to 40°C. The gas formed in the reaction is passed through a wash bottle filled with tetralin (also predried and distilled) in order to eliminate small quantities of  $Br_2$ , and through a trap cooled to  $-60^{\circ}C$  in order to remove the last traces of moisture. It is then frozen in a second trap cooled with liquid nitrogen. After completion of the reaction, the receiver trap is separated from the gas generating apparatus by melting the connection.

A more effective method for removal of the last traces of water involves trapping at  $-70^{\circ}$ C instead of  $-60^{\circ}$ C, so that some liquid HBr can accumulate through which the remaining HBr gas will bubble.

The condensed HBr is purified by subliming part of the solid product and collecting the middle fraction in a receiver cooled with liquid nitrogen. The container is then sealed off. The pressures in the preparation and fractionation sections of the apparatus should be monitored by means of an attached manometer. (For suitable apparatus, see Part I, p. 66 ff.)

$$\begin{array}{c} H_2 + Br_2 = 2 \, HBr \\ 2 \, 159.9 \, 161.9 \end{array}$$

The arrangement shown in Fig. 145 is used; hydrogen bubbles through wash bottle A, serving as a flowmeter. It then accumulates



Fig. 145. Preparation of hydrogen bromide.

in flask B, to which  $Br_2$  can be added in drops from dropping funnel C. The connecting tube from A must reach to the bottom of B. Between A and B a part of the  $H_2$  stream is diverted to C so that when the closed-off vessel C is depleted, pressure equilibrium will be maintained. The  $H_2$  stream carrying the  $Br_2$  vapor then enters Pyrex tube D (40-50 cm. long, 2-4 cm. in diameter), filled with platinized asbestos or platinized silica gel held in place by glass wool plugs. This tube is heated in electric furnace E. The tube is connected to tube F, which contains red phosphorus dispersed on glass spheres or Raschig rings, and to a wash bottle G,

п.

which contains a few milliliters of water to remove entrained phosphorus compounds. The HBr-H<sub>2</sub> mixture finally passes through a drying tube H filled with CaCl<sub>2</sub> (CaBr<sub>2</sub> is better, of course) and is frozen in trap J by cooling with liquid nitrogen.

**Procedure:** Before adding the  $Br_2$  to B, the air in the apparatus is displaced by a stream of  $H_2$ . When this has been accomplished, the furnace is heated to 350°C and the first portion of about 50 ml. of  $Br_2$  is admitted to container B. The  $H_2$  should pass through the bromine layer (25°C) in a rather fast stream in order to assure a constant excess of  $H_2$ . Deterioration of the catalyst may be recognized by the increased presence of free  $Br_2$  in the part of the apparatus connected to tube D. Care should be exercised to avoid channeling of the gas through tube D due to shrinkage of the catalyst. If no such precautions are taken, the  $H_2$ - $Br_2$  mixture is likely to emerge unconverted from the reactor.

The HBr frozen out in  $\mathcal{J}$  is purified by fractional distillation as indicated in method I.

As a safety measure it is desirable that container B not be exposed to direct light. It is best to paint B black (leaving some peepholes in order to be able to check the amount of  $Br_2$  present). When needed, rubber stoppers (which then must be frequently changed) and rubber tubing over the glass-to-glass connections may be used. Ground glass or fused joints are better, however. Because of the necessary pressure in the system, the stoppers on the wash bottles should be correspondingly well secured.

# Other preparative method:

III. Dehydration of concentrated HBr solutions with  $P_2O_5$ . A round-bottom flask is partly filled with very pure  $P_2O_5$ ; HBr solution is then added in drops from a dropping funnel, with simultaneous cooling. Purification of the gas stream thus produced is carried out as described in method I (A. Klemenc).

#### PROPERTIES:

Formula weight 80.93. Colorless gas. M.p.  $-87^{\circ}$ C, b.p.  $-67^{\circ}$ C; d.  $(-68^{\circ}$ C) 2.17. A saturated solution in H<sub>2</sub>O at 0°C contains 68.85% and at 25°C, 66% HBr. The constant-boiling acid at 760 mm. and 126°C contains 47.8% HBr.

Completely dry HBr may be stored for some time over Hg. After a while, decomposition sets in, possibly promoted by light and stopcock grease.

IV. Aqueous solutions of HBr may be prepared using the HBr prepared and purified according to I or II. If a less pure product is sufficient, it is possible to simplify the procedure in the following ways:

In method I: The HBr, after passing through the wash bottle containing tetralin, is led directly into water cooled with an ice-salt bath. The yield in this case is 94% of theoretical. When it is remembered that half the bromine is lost by combining with the tetralin, the yield based on total bromine reacted is 47%.

An even simpler method is to mix equal quantities of tetralin and water and then slowly drop  $Br_2$  in with continuous stirring. The aqueous and nonaqueous layers are separated in a separatory funnel, the nonaqueous layer is again washed with  $H_2O$ , and the wash water is combined with the main HBr solution.

In method II: Drying tube H and trap J are replaced by one or more interconnected wash bottles containing water and cooled by an ice-salt bath. In this case, approximately 65% HBr solutions are obtained.

v.

 $H_2SO_4 + KBr = KHSO_4 + HBr$ 98.1 119 136.2 80.9

A direct method, which is suitable only for the preparation of constant-boiling HBr solutions, depends on the effect of dilute sulfuric acid on KBr (concentrated  $H_2SO_4$  would oxidize the HBr to  $Br_2$ ).

A mixture of 120 g. of pulverized KBr and 200 ml. of  $H_2O$  is chilled with cold water and slowly reacted with 90 ml. of concentrated  $H_2SO_4$ . The temperature should not be allowed to rise above 75°C to retard possible formation of free bromine. The solution is then cooled to room temperature and the KHSO<sub>4</sub> is filtered off through a Büchner funnel (using hard filter paper). The filtrate is placed in a 500-ml. distillation flask equipped with a suitable condenser and receiver and heated over a wire gauze. After distilling off the water, the fraction that is collected begins to boil 1° below the boiling point of the azeotrope [b.p. 122.5°C (740 mm.), 126°C (760 mm.)] and the distillation is stopped as soon as the temperature begins to drop. The yield is about 85%.

This acid may still contain about 0.01% H<sub>2</sub>SO<sub>4</sub>. Acid that is completely free of H<sub>2</sub>SO<sub>4</sub>—in the highest attainable concentration is obtained if collection of the distillate is begun 5° below the boiling point of the constant-boiling acid. This distillate is then redistilled, but only the fraction at the boiling point of the azeotrope is collected.

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# Hydrogen Iodide

#### HI

The choice of preparative method depends on whether anhydrous HI or an HI solution is required. Method I (preparation of anhydrous HI from  $H_2$  and  $I_2$ ) is quite suitable for the preparation of HI solutions, while method III ( $H_2S + I_2$ ) is limited to solutions, unless (in accordance with II) the highly concentrated aqueous solution is dehydrated with  $P_2O_5$ . Since HI solutions soon turn brown on standing (due to the formation of iodine by light and air), a method (IV) for regenerating such solutions is also given.

I.

$$\begin{array}{c} H_2 + I_2 = 2 \, \text{HI} \\ 2 \, 253.8 \, 255.8 \end{array}$$

Hydrogen is passed over I<sub>2</sub> contained in a 250-ml. Pyrex flask A (see Fig. 146) which can be heated. A Pyrex tube B, 90 cm. long and 1.8-2 cm. in diameter, is connected to the flask. If possible, this tube is fused on directly. If absolutely necessary it may be connected by a ground joint. However, in this case the joint is sealed on the outside with asbestos-waterglass mixture. The part of the tube nearest flask A is filled for a length of 10-20 cm, with platinized asbestos or a mixture of asbestos with Pt sponge which is then heated to 500°C in a furnace. The tube is followed by a U tube C containing Cal<sub>2</sub> to dry the HI, a U tube D with KI to remove the last traces of iodine, and a freezing trap E which is cooled to  $-78^{\circ}$ C. A P<sub>2</sub>O<sub>5</sub> tube F serves as protection against atmospheric moisture. It is advisable to provide a bypass tube for H<sub>2</sub>. The bypass hydrogen stream may then be used as a carrier gas to carry unreacted iodine, or iodine formed by decomposition of HI, from the empty part of tube B back to flask A. In order to do this, stopcock  $H_1$  is opened, and the two-way stopcock is turned to a position  $180^{\circ}$  from that shown in Fig. 146. After cooling the catalyst, the I<sub>2</sub> is heated with a Bunsen burner and sublimed in a stream of hydrogen, which carries it into A.

Preliminary treatment of starting materials: Cylinder  $H_2$  is, as usual, passed over a Pt catalyst and through a system of drying tubes. The purest available  $I_2$  is used; it is dried in a vacuum over  $P_2O_5$  and, in order to remove any remaining Cl and Br, is intimately mixed with 5% by weight of KI. For the platinized asbestos, see the section on Platinum Metals; about 3 g. of asbestos fibers is saturated with 7 ml. of 10% H<sub>2</sub>PtCl<sub>6</sub> solution; the damp mixture is evaporated to dryness with continuous stirring and the product is then heated to red heat.

**Procedure:** After A has been charged with  $I_2$ , the air in the apparatus is carefully displaced by  $N_2$ , following which  $H_2$  is passed through. (If  $H_2$  were to be admitted while the apparatus still contained oxygen, the catalyst could promote an explosive reaction of the hydrogen-oxygen mixture.) The catalyst is now heated. The  $I_2$  in flask A is heated just enough to produce very small quantities of  $I_2$  vapor in the part of tube B that extends beyond the catalyst. Experience shows that the correct temperature of the  $I_2$  is reached when the condensation zone in the iodine flask lies somewhat higher than the side arm. Some  $I_2$  also condenses in the connecting tube between A and the catalyst and must therefore be carefully sublimed from time to time with a Bunsen burner. (In general, care should be taken during the entire run to assure that there are no solid  $I_2$  plugs at any point of the apparatus.) The HI product,



Fig. 146. Preparation of hydrogen iodide.

after passing through purification tubes C and D (which may be omitted if a high purity product is not required), is frozen in Eand then repeatedly fractionated. At the end of the run, the catalyst is cooled in a stream of  $H_2$ .

II. Another preparative method for anhydrous HI consists in dehydration of highly concentrated HI solutions by  $P_2O_5$ . A roundbottom flask is partially filled with very pure  $P_2O_5$ . The HI solution is then added in drops from a dropping funnel, while the flask is cooled. The gaseous product is dried in an adjoining tube with  $P_2O_5$ . Further purification of the HI product proceeds in accordance with method I.

#### PROPERTIES:

Formula weight 127.93. Colorless gas. M.p.  $-50.9^{\circ}$ C, b.p.  $-35.4^{\circ}$ C; d. (0°C) 5.66. Solid or liquid HI can be stored at a low temperature away from light. Solubility at 0°C, 900 g. of HI/100 g. of H<sub>2</sub>O.

No rubber tubing or stoppers should be used with HI, if at all possible. If greased stopcocks must be used, white vaseline is the most suitable lubricant.

III. Solutions of HI may be obtained if the product prepared according to I, instead of being condensed, is dissolved in water cooled in an ice-salt bath.

The following method may be used to prepare azeotropic aqueous HI solution:

H <sub>2</sub> S	$+ I_2 =$	2 HI	+ s
34.1	253.8	255.8	32.1

A suspension of 120 g. of I<sub>2</sub> in 150 ml. of H<sub>2</sub>O is vigorously stirred in a wide-neck 500-ml. vessel with a three-hole stopper (for a gas inlet tube which reaches below the surface of the liquid, a gas outlet tube, and a stirrer). The stirrer must fit the walls of the vessel as closely as possible. A stream of  $H_2S$  is then absorbed by the suspension, the flow rate being controlled so as not to exceed the absorption rate. Any slight excess of H<sub>2</sub>S leaves the reaction vessel through the gas outlet tube and goes either to a hood or is passed over the surface of a sodium hydroxide solution in a special flask; the outlet tube should not dip into the sodium hydroxide. After about an hour, the solution in the absorption flask becomes practically colorless due to the separation of considerable quantities of sulfur. The solution is then separated from the coarser sulfur particles by decantation and filtered through a glass frit to remove the fine sulfur. The H<sub>2</sub>S, still dissolved in the solution, is removed by a short period of boiling, after which a test of the solution should not give a reaction for sulfide.

The solution is distilled from a 250-ml. distillation flask, using boiling chips to avoid bumping. The fraction boiling from 125 to  $127^{\circ}$ C is collected. The yield is about 90% based on the I<sub>2</sub> used. The azetropic acid (57% HI) boils at 126°C (760 mm.), d 1.70. It fumes strongly in air. Aqueous HI solutions must be stored in dark, well-sealed flasks. It is advisable to seal the storage flasks with paraffin. As a further precaution against oxidation, the air above the surface of the liquid may be displaced by an inert gas before sealing the flask.

IV. Concentrated HI solutions that have become brown due to the separation of iodine may be regenerated as follows:

The reaction is carried out in a 500-ml. ground glass flask which is equipped with an inlet tube for inert gas  $(N_2, H_2 \text{ or } CO_2)$  and a fractionating column. The latter carries a dropping funnel (connected with a ground glass joint) on top. The iodine-containing HI solution is brought to near boiling with inert gas slowly passing through the flask. The hot solution is then reacted with 50% H<sub>3</sub>PO<sub>2</sub> solution, added by drops until decolorization occurs (only a few milliliters are needed for this, depending on the iodine content). The dropping funnel is now replaced by a ground joint thermometer, and the azeotropic acid is distilled off at  $125-127^{\circ}C$  (760 mm.).

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#### Ammonium lodide

#### NH₄I

I.

Powdered iodine (100 g.) is reacted with 280 ml. of 10% ammonia water (i.e., double the stoichiometric quantity) and 600 ml. of 3%  $H_2O_2$  (i.e., 33% excess). The  $I_2$  dissolves and  $O_2$  is evolved. In some cases, further  $H_2O_2$  solution must be added until the reaction mixture becomes pure yellow. The solution is evaporated on a steam bath.

The colorless crystals that separate deliquesce rapidly in moist air.

$$\begin{array}{rl} \mathrm{NH_{3}+HI}=\mathrm{NH_{4}I}\\ 17 & 127.9 & 144.9 \end{array}\\ (\mathrm{NH_{4})_{2}CO_{3}}\cdot\mathrm{H_{2}O}+2\,\mathrm{HI}=2\,\mathrm{NH_{4}I}+2\,\mathrm{H_{2}O}+\mathrm{CO_{2}}\\ 114.1 & 255.8 & 289.9 & 36 & 44 \end{array}$$

II.

A solution of HI and a solution of  $NH_3$  or  $(NH_4)_2CO_3$  are combined in stoichiometric quantities and evaporated until crystallization of  $NH_4I$  occurs.

To prepare completely iodine-free, colorless crystals (in a hydrogen atmosphere), see P. Wulff and H. K. Cameron, Z. phys. Chem. (B) 10, 350 (1930).

PROPERTIES:

Formula weight 144.96. Colorless, very deliquescent crystals. d. 2.56. Sublimes on heating. Solubility  $(25^{\circ}C)$ : 177 g./100 g. H<sub>2</sub>O.

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#### Potassium Iodide

#### KI

In order to prepare very pure KI, C.P. HI solution is allowed to react with  $KHCO_3$ , and the KI formed is heated in a stream of  $H_2$  to  $725^{\circ}C$  (m.p.  $680^{\circ}C$ ).

REFERENCES:

- I. I. Lingane and J. M. Kolthoff in: H. S. Booth. Inorg. Syntheses, Vol. I, New York-London, 1939, p. 163.
- J. M. Kolthoff and I. I. Lingane. J. Amer. Chem. Soc. <u>58</u>, 1524 (1936).

#### Iodine Monochloride

ICl

$$I_2 + Cl_2 = 2 ICl$$

$$253.8 \quad 70.9 \quad 324.7$$

About 300 ml. of cylinder  $Cl_2$  is condensed in a weighed 500-ml. flask surrounded by a Dry Ice-ether bath. Penetration of moisture into the flask must be avoided. Approximately half the stoichiometric quantity of  $I_2$  is added to the chlorine in the flask. The amount necessary is determined from the roughly estimated volume of  $Cl_2$  but should be weighed exactly before being added (300 ml. = 468 g. of  $Cl_2$  requires 1674 g. of  $I_2$ ; half = 837 g. of  $I_2$ ). The reaction mixture solidifies after addition of the  $I_2$ . The cold bath is removed, the flask is allowed to warm to room temperature, and the unreacted chlorine is thus removed by evaporation.

The flask and its contents are then weighed and, after subtracting the known weight of the empty flask and of the iodine added, the weight of  $Cl_2$  reacted is obtained. This quantity is always larger than that corresponding to the formation of ICl with a given quantity of  $I_2$ , indicating that some ICl<sub>3</sub> has formed. Therefore, iodine equivalent to the excess  $Cl_2$  is added.

The flask is closed with a glass stopper and allowed to stand 24 hours or longer at room temperature. The crude product (at least 1070 g.) is "recrystallized" once or twice for complete purification: the liquid ICl is cooled until about 80% of the material solidifies. The liquid portion is then discarded.

PROPERTIES:

Formula weight 162.38. Reddish-brown liquid at ordinary temperatures; exists in two solid modifications:  $\alpha$ -ICl, ruby-red needles (m.p. 27.19°C);  $\beta$ -ICl, brownish-red plates (m.p. 13.9°C), labile form.

The boiling point at atmospheric pressure cannot be determined exactly since ICl decomposes at the boiling point into  $I_2$  and  $Cl_2$ ; however, it lies in the vicinity of 100°C. d. (liq.) (29°C) 3.10.

Vigorously attacks cork, rubber and the skin, forming very painful patches (antidote: 20% hydrochloric acid).

Not hygroscopic; however,  $I_2O_5$  is formed on the vessel walls as a result of hydrolysis by the moisture of the air.

**REFERENCES:** 

J. Cornog and R. A. Karges. J. Amer. Chem. Soc. 54, 1882 (1932).

W. Stortenbeker. Rec. Trav. chim. Pays-Bas 7, 158 (1888).

J. Cornog and R. A. Karges in: H. S. Booth. Inorg. Syntheses, Vol. I, New York-London, 1939, p. 165.

# lodine Monobromide

IBr

I + Br = IBr126.9 79.9 206.8

A weighed quantity of finely powdered iodine is reacted in a cooled, round-bottom flask with the stoichiometric quantity of

dry bromine (added in portions). The mixture is then heated at  $45^{\circ}$ C in a nitrogen stream for a few hours. Further purification is achieved by allowing the melt to cool slowly (in the absence of moisture), and after the material has crystallized, most of the remaining liquid is decanted and discarded. The flask contents are remelted and the process repeated several times.

The product is best stored under dry  $N_2$  in a sealed container. Rubber stoppers should be avoided under any circumstances. It is best to work with IBr in closed systems since it attacks the eyes and mucous membranes rather vigorously.

#### PROPERTIES:

Brownish-black crystals with an odor similar to bromine. M.p. 40-41°C, b.p. 119°C; d. (0°C) 4.416, (50°C) 3.73. The vapor is largely dissociated.

REFERENCE:

V. Gutmann. Mh. Chemie 82, 156 (1951).

# Iodine Trichloride

FCl<sub>3</sub>

 $I_2 + 3 Cl_2 = 2 ICl_3$ 253.8 212.8 466.6

I. Since passing  $Cl_2$  over  $I_2$  gives impure products and poor yields, the method of Thomas and Depuis is recommended. In this procedure, iodine is added to excess liquid  $Cl_2$ , and the excess  $Cl_2$  is then evaporated.

A 200-ml. quantity of  $Cl_2$  (10% excess) is condensed in a flask cooled by a Dry Ice-acetone bath and protected from moisture. Finely powdered  $I_2$  (338.3 g.) is gradually added, whereupon orange ICl<sub>3</sub> immediately precipitates. To complete the reaction, the mixture is allowed to stand in a cooling bath for a few hours. The excess chlorine is distilled at room temperature into a second cooled container (where it may be reacted with more  $I_2$ ). The yield of ICl<sub>3</sub> is quantitative (622 g.).

II. According to E. Birk,  $Cl_2$  gas is passed over  $I_2$ , which is cooled by a Dry Ice-acetone bath to  $-79^{\circ}C$ , until yellow droplets of excess  $Cl_2$  are visible. The reaction mixture is allowed to remain in the cooling bath for a few hours and the  $Cl_2$  is then evaporated at room temperature. The yield is theoretical.

III. According to G. Mann, a layer of 500 g. of powdered iodine is spread over 250 g. of finely powdered  $KClO_3$  contained in a 1500-ml. Erlenmeyer flask. Then 250 ml. of water is added. Finally, 950 ml. of concentrated hydrochloric acid is added in small portions over a period of 45 minutes. The temperature should remain below 40°C. The cold solution is filtered through fritted glass; the  $ICl_3$  crystals are recrystallized from alcohol and dried over  $CaCl_2$  in vacuum. The yield, based on the  $I_2$  used, is 75%.

PROPERTIES:

Formula weight 233.3. Loose, orange powder or long, yellow needles with a penetrating, pungent odor. Very corrosive to the skin and leaves painful brown patches.

M.p. 101°C under the pressure of its own vapor (16 atm.). Very volatile even at room temperature and must therefore be stored in well-sealed flasks. Vapor pressure 1 atm. at 64°C. The vapor is almost completely dissociated to ICl and Cl<sub>2</sub>; at 77°C, dissociation to ICl and Cl<sub>2</sub> is complete. d.  $(-40^{\circ}C)$  3.203.

Used as a chlorinating agent and as an oxidant (e. g., in sulfide analysis), in the form of a 25-35% solution of ICl<sub>3</sub> in concentrated hydrochloric acid.

**REFERENCES:** 

- I. V. Thomas and P. Depuis. Compt. Rend. Hebd. Séances Acad. Sci. <u>143</u>, 282 (1906).
  H. S. Booth and W. C. Morris in: H. S. Booth. Inorg. Syntheses, Vol. I, New York-London, 1939, p. 167.
  II. E. Birk. Angew. Chem. 41, 751 (1928); Z. anorg. allg. Chem.
- E. Birk. Angew. Chem. <u>41</u>, 751 (1928); Z. anorg. alig. Chem. <u>172</u>, 399 (1928).
   E. Wilke-Dörfur and E. A. Wolff. Z. anorg. alig. Chem. <u>185</u>, 333 (1930).
- III. G. Mann. Magyar Kémiai Folyóirat <u>57</u>, 143 (1951); abstract in Chem. Zentr. <u>1953</u>, 349.

# **Polyhalides**

Numerous compounds of this sort are known. The selection given here—with the exception of  $KI_3 \cdot H_2O$  and  $HICl_4 \cdot 4H_2O$ —is so chosen that to each of the previously described interhalogen compounds there corresponds a polyhalide which yields that compound on decomposition.

# Potassium Triiodide

#### $KI_3 \cdot H_2O$

The theoretical quantity of  $I_2$  is added to a hot, saturated solution of KI; after the iodine dissolves, the mixture is cooled to  $0^{\circ}$ C, whereupon KI<sub>3</sub> · H<sub>2</sub>O crystallizes out.

PROPERTIES:

Dark brown, hygroscopic prisms which melt in a sealed tube at 38°C and liberate iodine at 225°C, leaving KI.

For a discussion of the fact that anhydrous  $KI_3$  is unstable at room temperature while the monohydrate is stable, see the references given under II.

REFERENCES:

- I. H. L. Wells and H. L. Wheeler. Z. anorg. allg. Chem. <u>1</u>, 453 (1892).
- II. N. S. Grace. J. Chem. Soc. (London) <u>1931</u>, 608.
   H. W. Foote and W. C. Chalker. J. Amer. Chem. Soc. <u>39</u>, 565 (1908).

# Cesium Dichlorobromide

#### CsBrCl<sub>2</sub>

I.

CsCl	$+ 1/_2 Br_2$	$+ \frac{1}{2} Cl_2$	$= C_s Br Cl_2$
168.4	79.9	35.5	283.8

A solution of 16.9 g. of CsCl in 85 ml. of water is prepared and treated with 8 g.  $Br_2$ . The solution is then slightly heated in order to hold in solution the CsClBr<sub>2</sub> that is formed. The solution is then saturated with Cl<sub>2</sub>; glittering yellow crystals of CsBrCl<sub>2</sub> form. These are filtered, washed with some water, and recrystallized from a small amount of water. A better yield is obtained if the CsCl is dissolved in only 45 ml. of water. The bromine is then added, red crystals of CsClBr<sub>2</sub> precipitate, and Cl<sub>2</sub> is then introduced without producing any harmful effects.

II. 
$$C_sBr + Cl_2 = C_sBrCl_2$$

According to Ephraim,  $CsBrCl_2$  may also be produced by introducing  $Cl_2$  into CsBr solution until saturation. Cremer and Duncan carried out the same reaction, but used dry CsBr at room temperature.

PROPERTIES:

Glittering yellow crystals which melt in a sealed tube at  $205^{\circ}$ C but which, when heated in the open at about  $150^{\circ}$ C, evolve bromine, leaving CsCl. (If the salt is not stored in well-sealed flasks, an appreciable amount of halogen is given off even at room temperature.)

**REFERENCES:** 

- I. H. L. Wells. Amer. J. Sci. [3] <u>43</u>, 28 (1892); Z. anorg. allg. Chem. 1, 98 (1892).
- II. F. Ephraim, Ber. dtsch. chem. Ges. <u>50</u>, 1083 (1917).
   H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London) <u>1931</u>, 1865; 1933, 187.

#### Potassium Dichloroiodide

KICl<sub>2</sub>

I. DRY PROCESS:

$$\begin{array}{c} \text{KIBr}_2 + \text{Cl}_2 = \text{KICl}_2 + \text{Br}_2 \\ \text{325.8} \quad 70.9 \quad \text{236.9} \quad 159.8 \end{array}$$

Dry  $Cl_2$  is allowed to react with dry  $KIBr_2$  at room temperature. After a few minutes  $KICl_2$  is formed and the  $Br_2$  produced is carried off in the  $Cl_2$  stream. (When the reaction is continued for a longer period,  $KICl_4$  is formed instead.)

It is also possible to prepare KICl<sub>2</sub> in a dry process by grinding KICl<sub>4</sub> with KIBr<sub>2</sub>:

$$KICl_4 + KIBr_2 = 2 KICl_2 + Br_2$$
  
307.8 325.8 473.8

and driving off the Br<sub>2</sub> formed as a byproduct.

**II. AQUEOUS PROCESS:** 

$$\begin{array}{c} \mathrm{KI} + \mathrm{Cl}_{2} = \mathrm{KICl}_{2} \\ 166 \quad 70.9 \quad 236.9 \end{array}$$

Chlorine is introduced into a very concentrated solution of KI until the initially precipitated  $I_2$  redissolves. In order to prevent

further chlorination to  $\text{KICl}_4$ , finely pulverized KI is added until the  $I_2$  that separates is redissolved—with slight heating if necessary. Crystallization occurs on cooling.

PROPERTIES:

Long, orange crystals, very unstable in air. Begins to soften at  $60^{\circ}$ C in a sealed tube; liberates the labile halogen at  $215^{\circ}$ C.

REFERENCES:

- I. H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London) <u>1931</u>, 1863.
- II. F. Ephraim. Ber. dtsch. chem. Ges. 50, 1086 (1917).

# Cesium Dichloroiodide

### CsICl<sub>2</sub>

A solution of 16.8 g. of CsCl in 170 ml. of water is prepared and, after addition of 2.7 g. of  $I_2$ , is brought almost to boiling. Chlorine is introduced into the hot solution until the  $I_2$  dissolves. An excess of  $Cl_2$  should be avoided to prevent formation of CsICl<sub>4</sub>. On cooling, CsICl<sub>2</sub> crystallizes out. It may be purified, if necessary, by recrystallization from a small amount of hot hydrochloric acid (1:1) and washing with a small amount of cold hydrochloric acid.

PROPERTIES:

Orange crystals which melt at  $238^{\circ}$ C in a sealed tube, evolving labile halogen at  $290^{\circ}$ C. More stable than KICl<sub>2</sub>.

REFERENCE:

H. L. Wells. Z. anorg. allg. Chem. 1, 96 (1892).

# Potassium Dibromoiodide

KIBr<sub>2</sub>

 $\begin{array}{rrrr} KI \ + \ Br_2 \ = \ KIBr_2 \\ 166 \ 159.8 \ 325.8 \end{array}$ 

Since KIBr<sub>2</sub> crystallized from aqueous solutions always contains water of crystallization, it must be prepared in a dry process. A given quantity of finely pulverized and dried KI is mixed with an equal quantity (by weight) of  $Br_2$  and the mixture allowed to stand in a sealed flask for three days. When the reaction ends, the product is freed from excess  $Br_2$  by placing the unstoppered flask in a desiccator over  $I_2$  or NaOH.

PROPERTIES:

Shiny red crystals which melt at  $58^{\circ}$ C in a sealed tube, evolving labile halogen at  $180^{\circ}$ C.

REFERENCES:

H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London) <u>1931</u>, 1857. W. N. Rae. J. Chem. Soc. (London) 107, 1290 (1915).

#### **Cesium Dibromoiodide**

#### CsIBr<sub>2</sub>

Ι.

 $C_{sI} + Br_{2} = C_{sIBr_{2}}$ 259.8 159.8 419.6

Finely pulverized and dried CsI (26 g.) is mixed with about 17 g. of  $Br_2$  and allowed to stand in a closed flask for about three hours. The excess  $Br_2$  is removed by allowing the open flask to stand in a desiccator over  $I_2$  or NaOH.

II.  $\begin{array}{ccc} C_{S}Br &+ \frac{1}{2}I_{2} &+ \frac{1}{2}Br_{2} &= C_{S}IBr_{2} \\ 212.8 & 126.9 & 79.9 & 419.6 \end{array}$ 

A solution of 21.3 g. of CsBr in 213 ml. of water is prepared and treated with 12.7 g. of  $I_2$  and 8g. of Br<sub>2</sub>. On cooling, CsIBr<sub>2</sub> crystallizes out.

PROPERTIES:

Glistening red crystals, stable in air. Melt at 243 to 248°C in a sealed tube, evolving labile halogen at  $320^{\circ}$ C. More stable than KIBr<sub>2</sub>.

**REFERENCES:** 

- I. H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London) <u>1931</u>, 1860.
  - W. N. Rae. J. Chem. Soc. (London) <u>1915</u>, 1290.
- II. H. L. Wells. Z. anorg. allg. Chem. 1, 94 (1892).

# Potassium Tetrachloroiodide

### **KICl₄**

I. DRY PROCESS:

$$\begin{array}{rcrcr} \text{KIBr}_2 &+ 2 \text{ Cl}_2 &= \text{ KICl}_4 &+ \text{ Br}_2 \\ 325.8 & 141.8 & 307.8 & 159.8 \end{array}$$

Dry KIBr<sub>g</sub> (see p. 296) is placed in a flask equipped with a glass stopper carrying an inlet tube (almost touching the bottom of the flask) and a gas outlet tube. Dry  $Cl_2$  is passed through for some hours; this removes the byproduct  $Br_2$  as soon as formed. The yield of KICl<sub>4</sub> is quantitative. Within a few minutes after the chlorine is introduced, KICl<sub>2</sub> is formed. Reaction with further quantities of  $Cl_2$  to produce KICl<sub>4</sub> requires several hours.

A dry preparation process from KI and Cl<sub>2</sub> is described by W. N. Rae, J. Chem. Soc. (London) 1915, 1290.

The formation of a pure product in solution is questionable because of the following equilibrium:

$$\text{KICl}_4 + \text{Cl}_2 + 3 \text{H}_2\text{O} \rightleftharpoons \text{KIO}_8 + 6 \text{HCl}$$

so the compound is better prepared in a dry process.

The formation of iodate can be sharply suppressed by adding hydrochloric acid and avoiding an excess of chlorine.

**II. SOLUTION PROCESS:** 

$$KI + 2 Cl_2 = KI Cl_4$$
  
166 141.8 307.8

Concentrated KI solution is acidified with hydrochloric acid and chlorine is introduced. The weight increase should be controlled so as to avoid an excess of chlorine. The yield is 70%.

For preparation of  $\text{KICl}_4$  from KCl solution,  $I_2$  and  $\text{Cl}_2$ , see the references under III.

PROPERTIES:

Golden yellow needles; m.p.  $116^{\circ}C$  in a sealed tube; in air, evolve ICl<sub>2</sub> even at room temperature.

**REFERENCES:** 

- I. H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London) <u>1931</u>, 1864.
- II. E. Filhol. J. Pharm. Chim. 29, 457 (1939).
- III. H. L. Wells and H. L. Wheeler. Z. anorg. allg. Chem. 2, 257 (1892).
  M. Gutiérrez de Celis and E. Moles. An. Soc. Españ. Física Quim. 30, 542 (1932).

# Tetrachloroiodic Acid HICl<sub>4</sub> · 4 H<sub>2</sub>O

 $\begin{array}{l} \text{HCl} + \text{ICl}_{3} + 4 \text{ H}_{2}\text{O} = \text{HICl}_{4} \cdot 4 \text{ H}_{2}\text{O} \\ \text{36.5} \quad 233.3 \quad 341.8 \end{array}$ 

A solution of 20 g. of  $ICl_3$  in the stoichiometric quantity (6.9 ml.) of concentrated hydrochloric acid (d 1.19) is prepared and cooled to  $0^{\circ}C$ , whereupon  $HICl_4 \cdot 4H_2O$  precipitates out.

PROPERTIES:

Orange-yellow, deliquescent plates which vigorously attack the skin, paper, etc. Not very stable; decomposed by solvents.

REFERENCE:

H. W. Cremer and D. R. Duncan. J. Chem. Soc. (London), <u>1931</u>, 1865.

#### **Dichlorine Oxide**

 $Cl_2O$ 

 $\begin{array}{rrrr} 2 \ \text{HgO} \ + \ 2 \ \text{Cl}_2 \ = \ \text{Cl}_2 \text{O} \ + \ \text{HgO} \ \cdot \ \text{HgCl}_2 \\ 433.2 & 141.8 & 86.9 & 488.1 \end{array}$ 

I. According to Bodenstein and Kistiakowsky, chlorine dried with concentrated  $H_2SO_4$  is passed over HgO loosely packed in a U tube (300-350 mm. long, 12 mm. in diameter). Shallow layers of HgO should be alternated with small glass-wool plugs. In order to maintain a uniform temperature between 18 and 20°C, the U tube is placed vertically in a water bath. The product is passed through a  $P_2O_5$  tube and is then frozen in a vessel cooled with liquid nitrogen.

To obtain the best possible yield, the dry chlorine is mixed with dry air in a ratio of 1:2 to 1:3. The air flow rate can be measured in a bubble counter and, if necessary, can be regulated by intermediate stopcocks. To avoid any pressure in the apparatus, the Cl<sub>2</sub>-air mixture is aspirated over the HgO at about 400-600 mm. The system is protected against moisture by a trap cooled with liquid nitrogen or by a drying tube. About 15 g. of crude Cl<sub>2</sub>O is formed in 4-6 hours.
The  $Cl_2O$  is purified by distillation at atmospheric pressure. The first few milliliters are discarded. The remaining  $Cl_2O$  is of satisfactory purity except for the last few milliliters.

The HgO used in the reaction is prepared as follows: yellow HgO is precipitated from  $Hg(NO_3)_2$  solution with sodium hydroxide. It is then suction-filtered, washed, dried and heated to 200-250°C.

The only suitable stopcock lubricants are Kel-F or Teflon greases.

According to Wallace and Goodeve, the  $Cl_2O$  formed as described above may be purified by passing the crude condensate first over  $P_2O_5$  and then over precipitated but not heated HgO to remove any traces of unreacted  $Cl_2$ . It is then fractionated three times.

II. Solutions of  $Cl_2O$  in inert solvents. As an example, 3.55 g. of  $Cl_2$  is dissolved in 100 ml. of carbon tetrachloride, 12 g. (i.e., somewhat more than the theoretical 10.82 g.) of carefully dried HgO is added, and the slurry shaken for 1.5 hours in the dark. Then the HgO  $\cdot$  HgCl<sub>2</sub> and the excess HgO are filtered off through a fritted glass filter. The solution must be kept in the dark and, if possible, refrigerated with Dry Ice.

PROPERTIES:

Yellowish-brown gas, with strong, unpleasant odor; deep brown liquid. M.p.  $-116^{\circ}$ C, b.p.  $3.8^{\circ}$ C.

Dissolves easily in water (forming HClO). At  $0^{\circ}$ C, 1 vol. H<sub>2</sub>O dissolves more than 100 vol. Cl<sub>2</sub>O.

The material can only be stored as a liquid or as a solid below  $-80^{\circ}$ C. Explodes on mixing with organic materials.

**REFERENCES:** 

I. M. Bodenstein and G. B. Kistiakowsky. Z. phys. Chem. <u>116</u>, 372 (1925).

A. Klemenc. Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], 2nd ed., Vienna, 1948. R. Schwarz and H. Striebich. Z. anorg. allg. Chem. <u>224</u>, 30 (1935).

J. J. Wallace and C. F. Goodeve. Trans. Faraday Soc. 27, 649 (1931).

M. Schmeisser and F. Schmitz. Unpublished.

York-London, 1957, p. 158.

II. E. A. Moelwyn-Hughes and C. N. Hinshelwood. Proc. Roy. Soc. (London) (A) <u>131</u>, 179 (1931).
G. H. Cady in: T. Moeller. Inorg. Syntheses, Vol. V, New

300

## Chlorine Dioxide

## $ClO_2$

Ι.

$$3 \text{ KClO}_3 + 3 \text{ H}_2 \text{SO}_4 = 2 \text{ ClO}_2 + \text{HClO}_4 + \text{H}_2 \text{O} + 3 \text{ KHSO}_4$$
  
367.7 294.2 135 100.5 18 408.4

A mixture of 20 g. of KClO<sub>3</sub> and 60 g. of washed and calcined sand is prepared in a 200-ml. round-bottom flask equipped with a special dropping funnel and a gas outlet tube (Fig. 147). The mixture is cooled with ice, and ice-cold, concentrated  $H_2SO_4$  is slowly added by drops. The ClO<sub>2</sub> product, somewhat contaminated with Cl<sub>2</sub>, is aspirated over  $P_2O_5$  and condensed by cooling with liquid nitrogen. It is purified by fractionation. Only the middle fraction is used.

The above method carries a danger of explosion, and  $ClO_2$  may be prepared more safely if it is diluted by simultaneously formed  $CO_2$  as in method II:

II. 
$$2 \text{ KClO}_3 + 2 \text{ H}_2\text{SO}_4 + \text{H}_2\text{C}_2\text{O}_4 \cdot 2 \text{ H}_2\text{O} = 2 \text{ ClO}_2 + 2 \text{ CO}_2 + 245.1 \\ 196.2 \\ 126 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\ 135 \\$$

A mixture of 122 g. of KClO<sub>3</sub> (about one mole), 100 g. of C.P. oxalic acid ( $H_2C_2O_4 \cdot 2H_2O$ , 0.8 mole), and a chilled solution of



Fig. 147. Dropping funnel apparatus for preparation of chloride dioxide.

108 g. (57 ml., 1.1 moles) of concentrated  $H_2SO_4$  (d. 1.84) in 400 ml. of H<sub>2</sub>O is prepared in a 1.5-liter ground joint flask. The mixture is slowly heated on a steam bath, and the smooth stream of ClO<sub>2</sub> and CO<sub>2</sub> is passed through a P<sub>2</sub>O<sub>5</sub> drying tube and into a receiver cooled to  $-78^{\circ}C$ . When a sufficient quantity of liquid ClO<sub>2</sub> has collected, a stopcock located between the  $P_2O_5$  tube and the receiver is closed. The receiver is then cooled to -110°C and the remaining CO<sub>2</sub> removed by suction. Final purification is by fractionation, retaining only the middle fraction.

III. Other preparative methods have been reported: from solid NaClO<sub>2</sub> and Cl<sub>2</sub> (Hutchinson and Derby) and from  $AgClO_3$  and  $Cl_2$  (King and Partington).

PROPERTIES:

Formula weight 67.46. Yellow gas. M. p.  $-59^{\circ}$ C, b.p.  $9.9^{\circ}$ C. The liquid is reddish brown, the solid is yellowish red. At  $-80^{\circ}$ C the vapor pressure is almost zero. d (gas) 3.09, d (liq.) ~ 1.5. IV. Solutions of ClO<sub>2</sub> in H<sub>2</sub>O may be prepared by passing the ClO<sub>2</sub>-CO<sub>2</sub> mixture produced in method II through water cooled by an ice-salt mixture. The ClO<sub>2</sub> is twenty times more soluble than CO<sub>2</sub>. One vol. of H<sub>2</sub>O at 4°C dissolves 20 vol. of ClO<sub>2</sub> (5.7%).

V. 
$$2 \operatorname{NaClO}_2 \cdot 3 \operatorname{H}_2 O + \operatorname{Cl}_2 = 2 \operatorname{ClO}_2 + 2 \operatorname{NaCl} + 6 \operatorname{H}_2 O$$
  
289 71 135 117 108

Aqueous solutions of  $ClO_2$  may also be prepared by passing a  $Cl_2$ -air mixture (maximum 5%  $Cl_2$ ) through two fritted glass wash bottles containing a 10% solution of sodium chlorite. Mixing of the two gases in the proper proportions is carried out by passing the  $Cl_2$  and air separately through two wash bottles filled with  $H_2O$ , adjusting the ratio by bubble counting, and combining the streams. The  $ClO_2$  formed in the chlorite solution is swept in a stream of air into two consecutive fritted glass wash bottles filled with water. These are cooled in an ice-salt mixture to aid the absorption. When the NaClO<sub>2</sub> solution in the first wash bottle changes suddenly from brown to a weak yellowish-green, it is exhausted and should be refilled. It is then used as the second wash bottle in the series.

General Precautions for Handling  $ClO_2$ . Gaseous or liquid  $ClO_2$  often explodes for no obvious reason. Stopcock grease on stopcocks and ground glass joints should be avoided; concentrated  $H_2SO_4$  or fluorocarbon grease should be used as lubricants. It is best to confine the dropping funnels to the type shown in Fig. 147.

Direct exposure to daylight should be avoided in preparation and storage of  $ClO_2$  because  $ClO_2$  decomposes easily on illumination. It is best to work in a darkened hood and to paint the outside of the containers black.

Chlorine dioxide attacks Hg; the Hg surface may be protected by a layer of  $H_2SO_4$  for a short time. Containers filled with solid or liquid  $ClO_2$  which are to be sealed off should be well cooled previous to sealing so that no gaseous  $ClO_2$  is present.

As a safety measure in case of breakage, coating the reaction vessels with an adherent plastic film is recommended.

The maximum safe partial pressure of  $ClO_2$  is 36 mm. (see Reference, under III).

REFERENCES:

I. M. Bodenstein, P. Harteck and E. Padelt. Z. anorg. allg. Chem. 147, 233 (1925).

- II. W. Bray. Z. phys. Chem. <u>54</u>, 569 (1906).
  H. J. Schumacher and G. Stieger. Z. phys. Chem (B) <u>7</u>, 364 (1930).
  E. Schmidt, E. Geisler, P. Arndt and F. Ihlow. Ber. dtsch. chem. Ges. <u>56</u>, 25 (1923).
- III. W. S. Hutchinson and R. J. Derby. Ind. Eng. Chem. <u>37</u>, 813 (1945).
  R. J. Derby and W. S. Hutchinson in: T. Moeller, Inorg. Syntheses, Vol. IV, New York-London, 1953, p. 154.
  F. E. King and J. R. Partington. J. Chem. Soc. (London) <u>1926</u>, 925.

## Dichlorine Hexoxide

## $Cl_2O_6$

Formation of  $Cl_2O_8$  by illumination of  $ClO_2$  and also by illumination of a mixture of  $Cl_2$  and ozone has been proven by Bodenstein, Harteck and Padelt.

According to Schumacher and Stieger,  $ClO_2$  and ozone react as follows:

$$2\operatorname{ClO}_2 + 2\operatorname{O}_3 = \operatorname{Cl}_2\operatorname{O}_6 + 2\operatorname{O}_2$$

Chlorine dioxide (see p. 301, method II), diluted with  $CO_2$ , is introduced into vessela (Fig. 148), cooled to  $-10^{\circ}C$ . Simultaneously,



The stopcocks must be greased with fluorocarbon lubricant because of the corrosive effect of  $Cl_2O_6$ , unless one prefers to use the glass valves suggested by Bodenstein.

When a sufficient quantity of the material has accumulated in a, the apparatus is sealed off at  $s_1, s_2$  and

 $s_3$ , and vessel *a* is evacuated through *v*, while being immersed first in finely crushed Dry Ice and then in an ice-water bath. About 2/3 of the liquid is allowed to boil off to assure that all the ClO<sub>2</sub> has been removed from the Cl<sub>2</sub>O<sub>6</sub>. If the Cl<sub>2</sub>O<sub>6</sub> is to be used immediately in a reaction, it is distilled from *a* into a reaction vessel attached at *v*.



hood

In the same way, but using break-seal valves, Goodeve and Richardson prepared  $Cl_2O_6$  of an especially high degree of purity.

PROPERTIES:

Formula weight 166.9. Deep red liquid. M.p.  $3.5^{\circ}$ C, b.p.  $203^{\circ}$ C (calcd.); d (3°C) 2.02. Vapor pressure (0°C) 0.31 mm.

Gaseous  $Cl_2O_6$  is largely dissociated to  $ClO_3$ ; the latter, even at room temperature, dissociates to  $ClO_2$  and  $O_2$  or to  $Cl_2$  and  $O_2$ . Liquid  $Cl_2O_6$  is considerably more stable. It is the least explosive of all the chlorine oxides; however, it does explode on contact with organic materials.

**REFERENCES:** 

- M. Bodenstein, P. Harteck and E. Padelt. Z. anorg. allg. Chem. 147, 233 (1925).
- H. J. Schumacher and G. Stieger. Z. anorg. allg. Chem. <u>184</u>, 272 (1929).
- C. F. Goodeve and F. D. Richardson. J. Chem. Soc. (London) 1937, 294.

## **Dichlorine Heptoxide**

 $\mathrm{Cl}_2\mathbf{O}_7$ 

 $\begin{array}{c} 2 \operatorname{HClO}_4 + \operatorname{P_2O_5} = \operatorname{Cl_2O_7} + 2 \operatorname{HPO_3} \\ 200.9 & 142 & 182.9 & 160 \end{array}$ 

I. Receiver a in the apparatus shown in Fig. 149, which contains about 30 g. of loose  $P_2O_5$  dispersed between Raschig rings, is cooled with liquid nitrogen for about 15 minutes before 4-5 ml. of 70% HClO<sub>4</sub> is added in drops from the dropping funnel. The tube of the funnel is bent so that it terminates in the vicinity of the cold wall of the receiver. In this way the heat of reaction is rapidly carried away and explosions are avoided. After completion of the addition of acid, the temperature of a is raised to  $-70^{\circ}$ C, and after ten minutes to  $-25^{\circ}$ C; the reaction mixture is allowed to stand for two hours at this temperature. The viscous mass in ais finally allowed to stand at  $0^{\circ}$ C for a long time to complete the reaction.

After reevacuating, a is slowly heated on a water bath to 90°C. The fraction that distills below 40°C is discarded: beginning with a bath temperature of 40°C, colorless liquid droplets are obtained in the liquid-nitrogen-cooled receiver which is attached to drying tower b (filled with  $P_2O_5$  dispersed between Raschig rings). According to Goodeve and Powney, lower chlorine oxides may be decomposed by passing the product gas over CuO wire, freshly



Fig. 149. Preparation of dichlorine heptoxide. reduced to copper and placed in a copper tube. However, the formation of lower chlorine oxides can be avoided in advance, according to J. J. Manley [J. Chem. Soc. (London) <u>121</u>, 331 (1922)], if before the reaction ozone is passed for 0.5 to 3 hours over the  $P_2O_5$  to be used. The temperature should be 175-200°C; the treatment oxidizes the lower phosphorus oxides which are responsible for the formation of the lower chlorine oxides. Excess ozone is then displaced with  $O_2$ .

If the  $Cl_2O_7$  product is not colorless, it is fractionated in vacuum.

II. According to Meyer and Kessler,  $Cl_2O_7$ may be successfully prepared—although with a yield of only 10%—by mixing one part by weight of anhydrous  $HClO_4$  (e.g., 20 to 30 g.) and 3 to 4 parts by weight of pure calcined kieselguhr with a mixture of one part by weight of  $P_2O_5$  and one part by

weight of kieselguhr. The reaction flask must be cooled.

The  $Cl_2O_7$  is distilled off at 80-90°C (2 mm.) and collected in a receiver cooled to -78°C. The relatively high temperature of 80-90°C is necessary because kieselguhr absorbs considerable quantities of  $Cl_2O_7$ . For further purification, a fractionating column may be attached.

III. A solution of  $\text{Cl}_2\text{O}_7$  in  $\text{CCl}_4$  is prepared as follows: 50 g. of  $P_2\text{O}_5$  is suspended in 120 ml. of  $\text{CCl}_4$  in a 500-ml. round-bottom flask. The flask contents must be continually stirred and cooled to 0°C. Then 8.2 g. of a 70% aqueous solution of  $\text{HClO}_4$  is slowly added, drop by drop. Next, most of the  $\text{CCl}_4$  is distilled off at 0°C, using an aspirator, and the mixture is then heated for a while to 70-75°C at atmospheric pressure. It is then distilled at 80°C, whereupon a mixture of CCl<sub>4</sub> and Cl<sub>2</sub>O<sub>7</sub> goes over and is trapped in a receiver cooled to -20°C.

When this yellowish mixture is heated for a short time to  $80^{\circ}$ C,  $Cl_2$  and lower chlorine oxides volatilize and a colorless, approximately 7-8% solution of  $Cl_2O_7$  remains. If a more concentrated solution is required,  $P_2O_5$  is added to this mixture and the entire operation is repeated. A 20-25% solution may be obtained by repeating the procedure several times.

General Precautions for Handling  $Cl_2O_7$ . Dichlorine heptoxide is more stable than the other chlorine oxides; it will, however, explode on impact or in contact with a flame. The greatest source of danger in the preparation is not so much in handling the  $Cl_2O_7$  product, but rather in the manipulation of the anhydrous  $HClO_4$  used as a starting material. The use of rubber and organic materials must be avoided; stopcocks must be lubricated with  $H_2SO_4$  or  $H_3PO_4$ , or still better, with fluorinated hydrocarbon grease.

The synthesis must be planned so that only about 2 ml. of  $Cl_2O_7$  is produced at a time.

PROPERTIES:

Formula weight 182.91. Colorless, very volatile oil. M.p.  $-91.5^{\circ}$ C, b.p. 82°C; d (0°C) 1.86. Vapor pressure (0°C) 23.7 mm.

**REFERENCES:** 

- I. A. Michael and W. T. Conn. Amer. Chem J. 23, 445 (1900).
   M. Schmeisser and D. Lützow. Unpublished.
   C. F. Goodeve and J. Powney. J. Chem. Soc. (London) 1932, 2078.
- II. F. Meyer and H. G. Kessler. Ber. dtsch. chem. Ges. <u>54</u>, 566 (1921).

## **Bromine Oxides**

## BROMINE DIOXIDE, BrO<sub>2</sub>

I. The preparation is carried out by ozonizing bromine in Freon 11 (CFCl<sub>3</sub>) at a low temperature. A solution of 1 g. of bromine in about 50 ml. of Freon 11 is prepared in a refrigerated container. This solution is saturated with ozone at  $-50^{\circ}$ C for about 30 minutes. The ozone dissolves in the Freon, giving a blue color. The ozone flow is then shut off and the solution is left for about 30 minutes in the refrigerant until a slight precipitate forms. A stream of ozone (precooled to  $-78^{\circ}$ C in a glass spiral) is then passed through the solution until the bromine has reacted completely (about 5-7 hours). Moisture is kept out by means of a  $P_2O_5$  tube. The BrO<sub>2</sub> forms as a solid precipitate with the color of egg yolk. The ozone and Freon are distilled off in a high vacuum. The product is very pure.

II. According to Schwarz and Schmeisser,  $BrO_2$  may be prepared from  $Br_2$  vapor and  $O_2$  in a glow discharge, using a discharge tube cooled by liquid nitrogen, which also serves as a gas trap (compare Part I, p. 91, Fig. 74).

PROPERTIES:

Solid with the color of egg yolk. No definite melting point; decomposes to  $Br_2$  and  $O_2$  at temperatures in the region of  $0^{\circ}C$ . May explode if heated too rapidly.

**REFERENCES:** 

I. M. Schmeisser and K. Joerger. Angew. Chem. <u>71</u>, 523 (1959).
 II. R. Schwarz and M. Schmeisser. Ber. dtsch. chem. Ges. <u>70</u>, 1163 (1937).
 R. Schwarz and H. Wiele. J. prakt. Chem., N. S. <u>152</u>, 157 (1939).

DIBROMINE MONOXIDE, Br<sub>2</sub>O

Slow heating of  $\text{BrO}_2$  (which is stable at low temperatures) in high vacuum from  $-40^{\circ}$ C upward causes it to decompose to  $\text{Br}_2$ ,  $O_2$ , a white, unidentified oxide, and  $\text{Br}_2O$ , which can be frozen out as a brown substance.

By separating the byproducts at  $-55^{\circ}C$ , the Br<sub>2</sub>O may be obtained in the pure state.

REFERENCE:

R. Schwarz and H. Wiele. J. prakt. Chem., N. S. 152, 157 (1939).

## **Diiodine Pentoxide**

### $I_2O_5$

Diiodine pentoxide may be prepared by thermal dehydration of iodic acid.

 $\begin{array}{c} 2 \, \mathrm{HIO_3} = \mathrm{I_2O_5} + \, \mathrm{H_2O} \\ 351.8 & 333.8 & 18 \end{array}$ 

Finely powdered  $\text{HIO}_3$  is heated in a stream of dry air for several hours at a temperature of 240-250°C in a glass tube which is placed in an aluminum block or an electric furnace in order to maintain the temperature as precisely as possible.

In order to obtain a colorless or, at most, a pale pink product, which corresponds as far as possible to the composition  $I_2O_5$ , the following precautions should be kept in mind:

According to Lamb, Bray and Geldard, the  $HIO_3$  produced from  $I_2$  and  $HCIO_3$  (see  $HIO_3$ , method III) is more suitable for producing colorless  $I_2O_5$  than the  $HIO_3$  produced from  $I_2$  and  $HNO_3$  or from  $I_2$ ,  $H_2O_2$  and  $HNO_3$ .

Special attention should be given to the careful purification and drying of the air stream. According to Moles,  $H_2SO_4$  should be avoided. The air should be purified by passage through alkaline KMnO<sub>4</sub> solution and over solid KOH, Na wire and  $P_2O_5$ .

Decomposition of iodic acid begins at  $70^{\circ}$ C in accordance with the equation  $3\text{HIO}_3 = \text{HIO}_3 \cdot I_2O_5 + H_2O$ . This reaction is favored if some  $\text{HIO}_3 \cdot I_2O_5$  or  $I_2O_5$  is added in advance. Otherwise, rapid heating will cause the  $\text{HIO}_3$  to melt at  $110^{\circ}$ C, which results in transition to  $\text{HIO}_3 \cdot I_2O_5$ . At  $200^{\circ}$ C,  $\text{HIO}_3 \cdot I_2O_5$  begins to evolve water and decomposes to  $I_2O_5$ . At  $240-250^{\circ}$ C this reaction proceeds rapidly and thoroughly. Keeping the above facts in mind, Baxter recommends that  $\text{HIO}_3$  be first heated for an adequate time at  $100^{\circ}$ C (first dehydration stage) and then for one hour at  $240^{\circ}$ C (second dehydration stage).

The final product from the method of Lamb, Bray and Geldard still contains 0.2% moisture; Baxter's product, 0.002%.

#### PROPERTIES:

White, hygroscopic crystals. Decomposition to  $I_2$  and  $O_2$  begins at 275°C and is rapid at 350°C. d (25°C) 4.8.

#### **PROPERTIES:**

- E. Moles and A. Perez-Vitoria. Z. phys. Chem. (A) <u>156 a</u> (Bodenstein. Anniversary Volume 583, (1931).
- G. P. Baxter and G. St. Tilley, Z. anorg. allg. Chem. <u>61</u>, 293 (1909).
- G. P. Baxter. Z. anorg. allg. Chem. 70, 41 (1911).
- A. B. Lamb, W. C. Bray and W. J. Geldard. J. Amer. Chem. Soc. 42, 1644 (1920).

## Hypochlorous Acid

**HClO** 

I. 
$$2 [Cl_2 \cdot 6 H_2O] + HgO = 2 HClO + HgCl_2 + 11 H_2O$$
  
357.8 216.6 104.9 271.5 198

Chlorine hydrate (e.g., 300 g.) is shaken in a wide-mouth bottle for 15 minutes with 3/4 of its weight (e.g., 225 g.) of HgO (prepared from HgCl<sub>2</sub> by precipitation with NaOH and dried at  $300^{\circ}$ C). The semifluid mass is then vacuum distilled; distilling off 1/3 of the mixture and collecting the distillate in a receiver cooled to  $-20^{\circ}$ C yields a greenish-yellow, 25% solution of HClO. This solution may be stored for some time at low temperature; it decomposes immediately, however, at  $0^{\circ}$ C. II.

$$\begin{array}{c} \text{Cl}_{2}\text{O} + \text{H}_{2}\text{O} = 2 \text{ HClO} \\ 86.9 \quad 18 \quad 104.9 \end{array}$$

A solution of  $\text{Cl}_2\text{O}$  in  $\text{CCl}_4$ , cooled to  $0^\circ\text{C}$ , is mixed with water  $(0^\circ\text{C})$  in a separating funnel (the stopcock of which must under no circumstances be coated with stopcock grease) and shaken vigorously for three minutes. The  $\text{CCl}_4$  layer is removed; the aqueous solution contains HClO which is free of  $\text{Cl}_2$ .

PROPERTIES:

Can be stored only in aqueous solution and is in equilibrium with the anhydride  $Cl_2O$ . The latter can be extracted from the solution, using  $CCl_4$ , for example.

REFERENCES:

- I. S. Goldschmidt. Ber. dtsch. chem. Ges. 52, 753 (1919).
- II. G. H. Cady in: T. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 160.

## Sodium Hypochlorite

NaClO · 5 H<sub>2</sub>O

$$\begin{array}{c} 2 \text{ NaOH} + \text{Cl}_2 + 4 \text{ H}_2\text{O} = \text{NaClO} \cdot 5 \text{ H}_2\text{O} + \text{NaCl} \\ 80 \quad 70.9 \quad 72 \quad 164.5 \quad 58.4 \end{array}$$

Chlorine is led through a glass tube, widened at the end like a funnel, into an ice-cold solution of 50 g, of NaOH in 50 ml. of H<sub>2</sub>O contained in a brown, 350-ml., wide-neck flask. The mixture soon becomes a slurry and it must be repeatedly shaken in order to cause further Cl<sub>2</sub> to be absorbed. The progress of the reaction is periodically checked by determining the weight. After about three hours, the weight increase is 35 g.; this is about 80% of the quantity of chlorine (43.7 g.) needed for complete saturation. The reaction is now interrupted and the NaCl which has precipitated out is filtered off on a glass frit funnel cooled with an ice-salt bath. The filtrate must likewise be cooled with a freezing bath. since disproportionation to chloride and chlorate occurs on warming. The filtrate is then placed in a cold bath at  $-40^{\circ}$ C, in which it solidifes completely within half an hour. After slowly raising the temperature to  $-5^{\circ}$ C, the crystals of NaClO  $\cdot$  5H<sub>2</sub>O are filtered through a fritted glass filter, externally cooled with an ice-salt bath.

#### PROPERTIES:

Colorless crystals, melting at  $18^{\circ}$ C in their own water of crystallization. This melt decomposes easily. By dehydration in

vacuum (over concentrated  $H_2SO_4$ ) at the lowest possible temperature, largely anhydrous NaClO may be obtained; however, it tends to decompose explosively. Decomposed by the CO<sub>2</sub> of the air. Solubility (0°C) 29.3 g. of NaClO  $\cdot$  5H<sub>2</sub>O/100 g. H<sub>2</sub>O.

REFERENCES:

- A. Sanfourche and L. Gardent. Bull. Soc. Chim. France [4] <u>35</u>, 1089 (1924).
- Private communication of I. G. Farbenindustrie A.-G., Werk Rheinfelden.

## Sodium Hypobromite

## NaBrO · 5 H₂O

 $\begin{array}{l} Br_2 \,+\, 2\,NaOH \,+\, 4\,H_2O \,=\, NaBrO \cdot 5\,H_2O \,+\, NaBr \\ 160 \quad 80 \qquad 209 \qquad 103 \end{array}$ 

A 40% sodium hydroxide solution (438 g., 306 ml.) is stirred and cooled to  $-3^{\circ}$ C in a wide-neck, round-bottom flask closed with a three-hole rubber stopper (for a stirrer, dropping funnel and thermometer). Bromine (314 g., 100 ml., 90% of the quantit, theoretically needed for 175 g. of NaOH) is slowly added (1-2 drops per second) with constant stirring. The temperature of the reaction mixture is held between -8 and  $-3^{\circ}$ C. Insufficient cooling results in the formation of NaBrO<sub>3</sub>; cooling below  $-8^{\circ}$ C should be avoided to prevent slow solidification of the flask contents.

During the addition of bromine, NaBr  $\cdot 2H_2O$  separates out. After completion of the addition of the bromine, the mixture is allowed to stand for one hour at  $-8^{\circ}C$ ; it is then filtered from the easily filterable NaBr  $\cdot 2H_2O$  (about 180 g.) through a glass frit filter, the filtrate being collected in a suction flask cooled to  $-5^{\circ}C$ .

The orange filtrate, with a content of about 60 g. of NaBrO per 100 ml., is supersaturated with NaBrO.5H<sub>2</sub>O. In order to induce crystallization, it must be seeded with some crystals prepared as follows: a few milliliters of NaBrO solution in a test tube are cooled to  $-50^{\circ}$ C while the walls are scratched with a thermometer, whereupon the entire tube contents solidfy. By briefly warming the test tube with the hand, the thermometer with the adhering crystal mass can be removed. A small part of these crystals is used to seed a few milliliters of NaBrO solution in a test tube at  $-5^{\circ}$ C. The pure NaBrO.5H<sub>2</sub>O which crystallizes out is now used to seed the remaining solution, maintained at  $-3^{\circ}$ C. The abundant precipitate of NaBrO.5H<sub>2</sub>O obtained is in the form of well-formed needles. The yield from 100 ml. of the filtrate is about 55 g. of crude moist material containing about 78% NaBrO.5H<sub>2</sub>O, 14%

NaBr·2H<sub>2</sub>O, and 3% NaBrO<sub>3</sub>. The remainder is water. The crystal mass is separated on a glass frit and immediately recrystallized from 2% NaOH in order to obtain the maximum possible separation of NaBr and NaBrO<sub>3</sub>. About 30 ml. of base (at 20°C) is used for each 100 g. of crude material. The solution is then filtered and the filtrate cooled with an ice-salt mixture. The crystals that are filtered off (when damp, about 18 g. from 100 g. of crude crystals) are dried for 1-2 hours on a precooled (0°C) porous clay plate placed in a desiccator (0°C) filled with silica gel.

PROPERTIES:

Formula weight (NaBrO) 118.91. Yellow crystals, readily soluble in  $H_2O$ . May be stored at  $-20^{\circ}C$  in a closed weighing vessel for a few days with only slight loss of NaBrO; decomposition, with formation of NaBr and NaBrO<sub>3</sub>, begins immediately at 0°C and is complete within two days.

Analysis after drying for one or two hours on clay at  $0^{\circ}$ C; about 92% NaBrO.5H<sub>2</sub>O, 2% NaBr.2H<sub>2</sub>O, 1% NaBrO<sub>3</sub>, 5% moisture.

REFERENCE:

R. Scholder and K. Krauss. Z. anorg. allg. Chem. 268, 279 (1952).

## **Potassium Hypobromite**

KBrO · 3 H<sub>2</sub>O

 $Br_2 + 2 KOH + 2 H_2O = KBrO \cdot 3 H_2O + KBr$ 159.8 112.2 189 119

This may be prepared from concentrated potassium hydroxide solution and  $Br_2$  in a manner similar to that previously described for NaBrO·5H<sub>2</sub>O. One mixes 465 g. (300 ml.) of 53% potassium hydroxide solution (245 g. of KOH and 220 g. of H<sub>2</sub>O) and 314 g. (100 ml.) of  $Br_2$ . This solution (about 250 ml.) is filtered from precipitated KBr (at this point the solution contains about 70 g. of KBrO per 100 ml.), and 125 g. of KOH and 157 g. (50 ml.) of  $Br_2$  are added with stirring. The temperature is -5 to -10°C. The precipitated KBr is again filtered. The filtrate, which contains about 83 g. of KBrO per 100 ml., is cooled to -40°C and seeded with several crystals obtained by cooling a portion of the filtrate to -80°C. After a short time a substantial yield of long, yellow needles of KBrO·3H<sub>2</sub>O is obtained. The precipitate is filtered off and dried on a precooled clay plate for about 20 hours in a desiccator at  $-20^{\circ}C$ .

**REFERENCE:** 

R. Scholder and K. Krauss. Z. anorg. allg. Chem. 268, 279 (1952).

## Sodium Chlorite

## $NaClO_2 \cdot 3 H_2O$

One method of preparation, based on the reaction of  $ClO_2$  with alkalis in the presence of  $H_2O_2$ , is given below. Other methods use, for example,  $SO_2$  or  $Mn(OH)_2$  instead of  $H_2O_2$ , or start with  $ClO_2$  and metals or amalgams.

 $\begin{array}{c} 2\,\text{ClO}_2 + \text{Ba}(\text{OH})_2 \cdot 8\,\text{H}_2\text{O} + \text{H}_2\text{O}_2 = \text{Ba}(\text{ClO}_2)_2 + 10\,\text{H}_2\text{O} + \text{O}_2\\ 135 & 315.5 & 34 & 272.3 & 180 & 32\\ \end{array}$  $\begin{array}{c} \text{Ba}(\text{ClO}_2)_2 + \text{Na}_2\text{SO}_4 \cdot 10\,\text{H}_2\text{O} = 2\,\text{Na}\text{ClO}_2 \cdot 3\,\text{H}_2\text{O} + \text{Ba}\text{SO}_4 + 4\,\text{H}_2\text{O}\\ 272.3 & 322.2 & 289 & 233.4 & 72.1 \end{array}$ 

Chlorine dioxide is synthesized from 24.5 g. of KClO<sub>3</sub>, 20 g. of oxalic acid, 21.6 g. (11.8 ml.) of concentrated  $H_2SO_4$  (d 1.84) and 80 ml. of water, according to method II given in the section Chlorine Dioxide, and is introduced into an Erlenmeyer flask containing 200 ml. of ice-cold  $H_2O$ . The yellow-orange ClO<sub>2</sub> solution is shaken until it is decolorized with an excess of solid Ba(OH)<sub>2</sub>  $\cdot$  8H<sub>2</sub>O (i.e., with more than the theoretical quantity of 31.6 g., because of carbonate impurities present) and with 12 g. of 30%  $H_2O_2$ . The BaCO<sub>3</sub> precipitate is filtered off. The filtrate is boiled and treated with solid Na<sub>2</sub>SO<sub>4</sub> until the excess barium ion precipitates as BaSO<sub>4</sub>. The BaSO<sub>4</sub> is filtered and the solution evaporated on a steam bath until crystals of NaClO<sub>2</sub>  $\cdot$  3H<sub>2</sub>O separate. The yield is about 15.6 g. (54%).

PROPERTIES:

Formula weight 144.51. White, flaky crystals; may be dehydrated over KOH in a desiccator. Anhydrous  $NaClO_2$  explodes on impact.

**REFERENCES:** 

F. Foerster and P. Dolch. Z. Elektrochem. 23, 138 (1917). G. R. Levi. Atti Accad. Naz. Lincei [5] <u>31</u>, 214 (1922); Gazz. Chim. Ital. <u>52</u>, 418 (1922).

## Chloric Acid

HClO<sub>3</sub>

 $\begin{array}{c} Ba(ClO_3)_2 + H_2SO_4 = 2 HClO_3 + BaSO_4 \\ H_2O & 98,1 & 169 & 233.4 \\ 322.3 & \end{array}$ 

A solution of 322 g. of barium chlorate in 500 ml. of boiling water is prepared. A hot mixture of 98 g. of concentrated  $H_2SO_4$ 

(53.3 ml., d 1.84) and 53.3 ml. of  $H_2O$  is then slowly added with stirring. Care should be taken to assure that a small excess of  $Ba(ClO_3)_2$  rather than of  $H_2SO_4$  exists at the end of the addition. The  $BaSO_4$  precipitate is allowed to settle for at least one hour. Then 2/3 of the clear solution is poured off and the remainder filtered through a Büchner funnel. The filtrate is combined with the decanted solution, yielding about 660 ml. of a 22% solution of  $HClO_3$  (d 1.11). Evaporation of the solution in a vacuum desiccator over concentrated  $H_2SO_4$  produces concentrations up to 40%. (The 40% solution corresponds to the composition  $HClO_3 \cdot 7H_2O$ ; d 1.28.)

Alternate Method: This method uses cation-exchange resins to exchange metal ions (e.g.,  $Na^+$ ) for  $H^+$ .

According to Samuelson, this procedure is possible with  $\text{ClO}_3^-$ (e.g., in the form of NaClO<sub>3</sub>) while in the case of ClO<sup>-</sup>, BrO<sub>3</sub>, IO<sub>3</sub>, for example, the acid is reduced by the resin. Preparation of an approximately 10% HClO<sub>3</sub> solution, which then can be concentrated, may be carried out in accordance with a method described by Klement.

#### PROPERTIES:

Formula weight 84.46. Forms colorless solutions which may be stored in glass-stoppered bottles. Pure solutions undergo slight decomposition at  $95^{\circ}$ C; impure solutions decompose at as low as  $40^{\circ}$ C.

#### **REFERENCES:**

- A. B. Lamb, W. C. Bray and W. I. Geldard. J. Amer. Chem. Soc. 42, 1743 (1920).
- O. Samuelson. IVA. 17, 5 (1946).
- R. Klement. Z. anorg. allg. Chem. 260, 271 (1949).

## Ammonium Chlorate

## NH<sub>4</sub>ClO<sub>3</sub>

Ι.

$$\frac{2 \text{ KClO}_3 + (\text{NH}_4)_2 \text{SO}_4}{245.1 \quad 132.1 \quad 203 \quad 174.2}$$

The directions will be found below under  $Ba(ClO_3)_2 \cdot H_2O$ , since NH<sub>4</sub>ClO<sub>3</sub> is an intermediate product in that preparation. If NH<sub>4</sub>ClO<sub>3</sub> is to be isolated as such, the NH<sub>4</sub>ClO<sub>3</sub> solution, freed of alcohol, is evaporated before the addition of Ba(OH)<sub>2</sub> for crystallization (as described in the other procedure). Since the substance

thus prepared may still contain  $SO_4^2$ , it may be preferable to follow the entire procedure for  $Ba(ClO_3)_2$  and then prepare the desired NH<sub>4</sub>ClO<sub>3</sub> from the easily purified  $Ba(ClO_3)_2$  in accordance with method III.

п.

 $\begin{array}{c} HClO_3 + NH_3 = NH_4ClO_3\\ 84.5 & 17 & 101.5\\ 2 \ HClO_3 + (NH_4)_2CO_3 \cdot H_2O = 2 \ NH_4ClO_3 + CO_2 + 2 \ H_2O\\ 168.9 & 114.1 & 203 & 44 & 36 \end{array}$ 

A chloric acid solution is reacted with the stoichiometric quantity of  $NH_3$  or  $(NH_4)_2CO_3 \cdot H_2O$  and the solution placed in a desiccator over  $H_2SO_4$  to crystallize.

III.

 $\begin{array}{c} \text{Ba}(\text{ClO}_3)_2 + (\text{NH}_4)_2 \text{SO}_4 = 2 \text{ NH}_4 \text{ClO}_3 + \text{BaSO}_4 \\ \cdot \text{H}_2 \text{O} & 132.1 & 203 & 233.4 \\ 322.3 & & & \end{array}$ 

Concentrated solutions of the reagents are brought together in stoichiometric ratios; after filtering off the  $BaSO_4$ , the solution is evaporated.

PROPERTIES:

Formula weight 101.5. Small, needle-shaped crystals, which are unstable and therefore cannot be stored for any length of time. Caution should be exercised in handling  $\rm NH_4ClO_3$  since the substance occasionally explodes without apparent cause. It is definitely explosive at temperatures above  $100^{\circ}C$ . If the material is spread in a thin layer in the open, it may be manipulated without danger. It readily dissolves in water.

**REFERENCES:** 

- I. Vanino. Handb. d. präp. Chem., Inorganic Section, 2nd ed., Stuttgart, 1925, p. 459.
- II. I. W. Retgers. Z. phys. Chem. 5, 448 (1890).
- III. Ullmann. Enzyklopädie der techn. Chemie, 2nd ed., Berlin-Vienna, 1928/32, Vol. 3, p. 297.

**Barium Chlorate** 

 $Ba(ClO_3)_2 \cdot H_2O$ 

 $\begin{array}{c} 2 \ \text{KClO}_8 + (\text{NH}_4)_2 \text{SO}_4 = 2 \ \text{NH}_4 \text{ClO}_3 + \text{K}_2 \text{SO}_4 \\ 245.1 \ 132.1 \ 203 \ 174.2 \end{array}$ 

 $2 \text{ NH}_{4}\text{ClO}_{3} + \text{Ba}(\text{OH})_{2} \cdot 8 \text{ H}_{2}\text{O} = \text{Ba}(\text{ClO}_{3})_{2} \cdot \text{H}_{2}\text{O} + 2 \text{ NH}_{3} + 9 \text{ H}_{2}\text{O}$ 203 315.3 322.3 34 162

A mixture of 122.6 g. of  $KClO_3$ , 70 g. of  $(NH_4)_2 SO_4$  and 350 ml. of hot water is evaporated in a porcelain dish with constant stirring

until a thin slurry forms. After cooling, a fourfold quantity of 80% ethyl alcohol is added, resulting in the separation of insoluble  $K_2SO_4$  from the NH<sub>4</sub>ClO<sub>3</sub>. The  $K_2SO_4$  residue is filtered and washed several times with alcohol. The filtrate is freed of alcohol by distillation. The NH<sub>4</sub>ClO<sub>3</sub> residue (caution: NH<sub>4</sub>ClO<sub>3</sub> has a tendency to explode!) is reacted in a porcelain dish on a steam bath with a sufficient quantity of hot concentrated Ba(OH)<sub>2</sub> · 8H<sub>2</sub>O solution [at least 160 g. of Ba(OH)<sub>2</sub> · 8H<sub>2</sub>O dissolved in about 160 ml. of hot water] so that the ammonia odor disappears completely and the solution finally gives a definite alkaline reaction. It is then evaporated to dryness. The residue is dissolved in a fivefold quantity of H<sub>2</sub>O, and CO<sub>2</sub> is bubbled through the solution until the precipitation of BaCO<sub>3</sub> is completed. The BaCO<sub>3</sub> is filtered off and the solution evaporated to crystallization.

PROPERTIES:

Colorless, columnar prisms. M.p. (anhydrous salt)  $414^{\circ}$ C; d 3.18. Solubility (0°C): 27.4 g.; (100°C) 111.2 g./100 g. of H<sub>2</sub>O.

**REFERENCE:** 

Vanino. Handb. d. präp. Chem., Inorganic Section, 2nd ed., Stuttgart, 1925, p. 297.

## **Bromic Acid**

#### HBrO<sub>3</sub>

$$\begin{array}{c} Ba(BrO_{3})_{2} + H_{2}SO_{4} = 2 \ HBrO_{3} + BaSO_{4} \\ H_{2}O & 98 & 257.8 & 233.4 \\ 411.2 \end{array}$$

Finely powdered  $Ba(BrO_3)_2 \cdot H_2O(100 \text{ g.})$  is mixed with a cold solution of 15 ml. (27.6 g., i. e., 10% excess) of concentrated  $H_2SO_4$  (d 1.84) in 275 ml. of  $H_2O$ . The reaction flask is placed in an ice-salt bath during the addition. Following the addition, the flask is left for several hours in the bath and frequently shaken. The flask contents are then diluted by at least a factor of two; the exact amount of  $Ba(OH)_2 \cdot 8H_2O$  solution necessary to remove the excess  $H_2SO_4$  is added [7.75 g. of  $Ba(OH)_2 \cdot 8H_2O$  is needed for this], which causes further  $BaSO_4$  precipitation. The mixture is allowed to settle and the clear  $HBrO_3$  solution is then decanted; the  $BaSO_4$  is filtered off and the filtrate combined with the decanted solution. The acid may be concentrated to 50% by vacuum evaporation at as low a temperature as possible. PROPERTIES:

Formula weight 128.92. Colorless solution.

REFERENCE:

O. Burchard. Z. phys. Chem. 2, 814 (1888).

### **Barium Bromate**

 $Ba(BrO_3)_2 \cdot H_2O$ 

 $\begin{array}{r} 2 \text{ KBrO}_3 + \text{ BaCl}_2 \cdot 2 \text{ H}_2\text{O} = \text{Ba}(\text{BrO}_3)_2 \cdot \text{H}_2\text{O} + 2 \text{ KCl} + \text{H}_2\text{O} \\ 334 \quad 244.3 \quad 411.2 \quad 149.1 \quad 18 \end{array}$ 

A solution of 334 g. of  $\text{KBrO}_3$  in 700 ml. of boiling water is prepared; a hot solution of 244 g. of  $\text{BaCl}_2 \cdot 2\text{H}_2\text{O}$  in 400 ml. of  $\text{H}_2\text{O}$  is added. The mixture is cooled and the supernatant liquid is decanted. The residue is washed several times with 100-ml. portions of cold water and then suction-filtered. The yield is almost quantitative. For further purification the product may be recrystallized once or several times from boiling  $\text{H}_2\text{O}$ .

**PROPERTIES:** 

White crystals. M.p.  $260^{\circ}$ C (dec.); d. 3.99. Solubility ( $10^{\circ}$ C): 0.44 g.; ( $100^{\circ}$ C) 5.39 g./100 g. of H<sub>2</sub>O.

REFERENCE:

D. W. Pearce and R. G. Russel in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 20.

### Iodic Acid

#### HIO<sub>3</sub>

lodic acid may be prepared by oxidation of  $I_2$  with HNO<sub>3</sub> or with a mixture of HNO<sub>3</sub> and  $H_2O_2$ , but even with a clear reaction mixture a pure white product seldom results (I and II). Colorless HIO<sub>3</sub> is formed from  $I_2$  and HClO<sub>3</sub>, which in turn is prepared from Ba(ClO<sub>3</sub>)<sub>2</sub> and  $H_2SO_4$  (III). The procedure based on the reaction of Ba(IO<sub>3</sub>)<sub>2</sub> [prepared from Ba(ClO<sub>3</sub>)<sub>2</sub> and  $I_2$ ] with  $H_2SO_4$  is unsatisfactory, since it does not give a  $H_2SO_4$ -free product. I.

 $3I_2 + 10 HNO_3 = 6 HIO_3 + 10 NO + 2 H_2O$ 761.5 630 1055.5 300 36

Twice sublimed  $I_2$  (100 g.), in an Erlenmeyer flask covered by a water-cooled round-bottom flask, is heated to 70-80°C with pure fuming nitric acid until the solution becomes light yellow. The mixture is then evaporated to dryness on a steam bath, treated several times with some  $H_2O$ , and again evaporated to dryness. The residue is dissolved in concentrated nitric acid on a steam bath and the clear, colorless solution rapidly cooled in an ice bath. The crystals are suction-filtered on fritted glass and dried for several days in a desiccator over solid KOH. Large crystals may be obtained if a seeded HIO<sub>3</sub> solution in 20% HNO<sub>3</sub> is allowed to evaporate at room temperature or over CaCl<sub>2</sub> in a vacuum desiccator. The crystals are then filtered and washed with the minimum amount of water.

ΊΙ.

 $I_2 + 5 H_2 O_2 = 2 H I O_3 + 4 H_2 O_2 = 253.8 170 351.8 72$ 

Finely divided  $I_2$  (50 g.), which is best prepared by oxidation of an iodide solution with  $Cl_2$ ,  $H_2O_2$  or some other oxidizing agent, is placed in a 750-ml. flask and heated on a water bath (70°C) with 50 ml. of concentrated nitric acid, 25 ml. of 30%  $H_2O_2$  ( $H_2O_2$  free from organic stabilizers should be used) and 50 ml. of  $H_2O$ . A water-cooled, round-bottom flask is used to cover the reaction vessel. After repeated shaking, the reaction suddenly begins and the color fades. The addition of  $H_2O_2$  is continued until all the  $I_2$ present has reacted. The solution is then evaporated to dryness; the residue is redissolved in a minimum of water and treated as indicated below.

If organic material is present in the  $H_2O_2$ , the residue after evaporation is dark. If this is the case, the residue is heated two hours at 140-150°C and then for a while at 170-180°C; after cooling, the HIO<sub>3</sub> is extracted with a very small quantity of hot water, and this solution is filtered and crystallized.

Because of the great solubility of  $HIO_3$ , large losses occur during crystallization; therefore, aqueous  $HIO_3$  solutions may be mixed with an equal volume of concentrated nitric acid and evaporated to one third their volume. This usually results in separation of  $HIO_3$  even from hot solutions.

III. 
$$I_2 + 2 \text{ HClO}_3 = 2 \text{ HIO}_3 + \text{Cl}_2$$
  
253.8 168.9 351.8 70.9

lodine (100 g.) is mixed with a volume of  $HClO_3$  solution containing 68.55 g. (3% excess) of  $HClO_3$ . The reaction flask is equipped with an air inlet tube and an outlet tube to carry the  $Cl_2$  into an absorption solution (e.g., NaOH). The reaction mixture is heated and, after the reaction begins, air is slowly passed through. The reaction is completed in about 20 minutes. The solution is then cooled and filtered to remove impurities [e.g., small quantities of  $Ba(IO_g)_2$  from traces of barium ion in the  $HClO_3$ ]. The filtrate is evaporated to dryness in a dish, using vigorous agitation. It may also be recrystallized as described in methods I or II.

**PROPERTIES:** 

Formula weight 175.93. Colorless crystals. M.p.  $110^{\circ}C$  (conversion to HI<sub>3</sub>O<sub>8</sub>); d (0°C) 4.629. Water is partially eliminated even at 70°C, especially if even a trace of HIO<sub>3</sub>·I<sub>2</sub>O<sub>5</sub> is present. Above 220°C, complete dehydration to I<sub>2</sub>O<sub>5</sub> occurs.

Solubility (0°C): 286 g. HIO<sub>3</sub>/100 ml. H<sub>2</sub>O; (25°C): 141 g. HIO<sub>3</sub>/100 g. HNO<sub>3</sub> [d (25°C) 1.4].

Very readily soluble in water, but is not hygroscopic. Since  $HIO_3$  is light sensitive, it is best to carry out the reaction in complete darkness to obtain a colorless product.

### REFERENCES:

- I. E. Moles and A. Perez-Vitoria. Z. phys. Chem. (A) <u>156a</u> (Bodenstein Anniversary Volume), 583 (1931).
  G. P. Baxter and G. St. Tilley. Z. anorg. allg. Chem. <u>61</u>, 295 (1909).
- II. W. C. Bray and A. L. Caulkins, J. Amer. Chem. Soc. <u>53</u>, 44 (1931).

M. Guichard. Bull. Soc. Chim. France [4] 5, 723 (1909).

III. A. B. Lamb, W. C. Bray and W. J. Geldard. J. Amer. Chem. Soc. 42, 1643 (1920).

## Perchloric Acid

## **HClO₄**

$$\underset{138.5}{\text{KClO}_4} + \underset{100.4}{\text{H}_2\text{SO}_4} = \underset{100.4}{\text{HClO}_4} + \underset{100.4}{\text{KHSO}_4}$$

I. A fractionating flask equipped with a ground glass stopper is connected, either directly or through a ground glass joint, to a 75-cm.-long condenser, which is in turn connected to a receiver cooled to  $-40^{\circ}$ C (or, if sufficient, only to  $-20^{\circ}$ C). The latter is connected, through a tube filled with soda-lime, to an aspirator. The fractionating flask is charged with 25 g. of KClO<sub>4</sub> and 100 g. of  $H_2SO_4$  (d 1.84) and the contents slowly heated on a bath at a pressure of 10-20 mm. The flask should be immersed in the heating bath only to the liquid level so that the vapor space will not become overheated. The reaction begins at about 90°C; further heating is carried out at such a rate that a temperature of 160°C is reached in about one hour. The reaction mixture is allowed to remain for about two hours at this temperature. By this time all the KClO<sub>4</sub> has dissolved; the HClO<sub>4</sub> is then distilled.

The crude, yellowish distillate is immediately redistilled at  $35-40^{\circ}$ C and 10-20 mm. on a steam bath. It is advisable to use a ground joint boiling capillary in the second distillation, but this is not necessary for the first. Small quantities of  $ClO_2$ , which color the acid yellow, may be quickly and completely removed by passing dry air through the solution. Dark-yellow acid cannot be decolorized either by the passage of air or by vacuum distillation. II. Anhydrous acid may be prepared from commercial 70% aqueous perchloric acid solution by mixing it with a fivefold quantity of  $H_2SO_4$  (95.6%) and distilling at 90 to  $160^{\circ}$ C at 20-30 mm. If the pressure falls below this value, loss of  $HClO_4$  by volatilization occurs. Use the apparatus described in I.

General Precautions for Handling Anhydrous  $HClO_4$ . All connections in the apparatus, if not fused with a torch, must be made with ground glass joints lubricated with  $H_3PO_4$ ,  $H_2SO_4$  or  $HClO_4$ . Rubber stoppers and rubber tubing must not be used. If solid  $HClO_4 \cdot H_2O$  should form in the receiver during distillation—perhaps because of too rapid heating—the distillation should be stopped immediately. If carefully cleaned glassware and pure starting materials are used, the preparation of  $HClO_4$  is not at all dangerous. The free acid should not come in contact with wood, because this may result in an explosion. Explosions always occur when organic material comes into contact with the anhydrous acid.

The skin must be thoroughly protected from the anhydrous acid (painful, lingering wounds!).

The acid may be stored for a long time below  $0^{\circ}$ C without decomposition. Acid residues should never be disposed of by pouring into a waste bucket, but should be either discarded (in portions) in the open or diluted with a large amount of cold water. Caution is advisable even under these conditions!

## PROPERTIES:

Water-clear, mobile liquid which fumes slightly in the air at room temperature.  $d_4^{22}$  1.764. M.p. -112°C, b.p. 130°C (760 mm. extrapolated: the acid decomposes at about 90°C), 14°C (15 mm.), 16°C (18 mm.), 17.3°C (20 mm.), 39°C (56 mm.).

REFERENCES:

- I. D. Vorländer and R. von Schilling. Liebigs Ann. Chem. <u>310</u>, 369 (1900).
   A. Michael and W. T. Conn. Amer. Chem. J. <u>23</u>, 444 (1900).
   K. van Emster. Z. anorg. allg. Chem. <u>52</u>, 270 (1907).
  - H. J. van Wyk, Z. anorg. allg. Chem. 48, 4 (1906).
- II. E. Linde, Z. Elektrochem. 30, 255 (1924).
  K. Berger. Thesis, Leipzig, 1928; H. Distler, unpublished experiments, Freiburg i. Br.

## **Alkaline Earth Perchlorates**

## HYDROUS PERCHLORATE

The hydrous perchlorates  $Mg(ClO_4)_2 \cdot 6H_2O$ ,  $Ca(ClO_4)_2 \cdot 4H_2O$ ,  $Sr(ClO_4)_2 \cdot 4H_2O$  and  $Ba(ClO_4)_2 \cdot 3H_2O$  are prepared from the corresponding oxides, carbonates, chlorides or nitrates by dissolving these in slightly more than the theoretical quantity of 70% HClO<sub>4</sub> (in the case of oxides, a slight excess of the oxide is used and later filtered off) and concentrating the solution by evaporation. The crystals that separate out are centrifuged and dried in a desiccator.

## ANHYDROUS PERCHLORATES

These are prepared by heating the hydrous perchlorates to  $250^{\circ}$ C in a vacuum drying oven, at pressures ranging from 1 to 10 mm.

Another method is based on the reaction of solid alkaline earth carbonates with solid  $NH_4ClO_4$  (the reactants are ground together in a ball mill) at 250°C and at pressures ranging from 1 to 10 mm., according to the following equation:

$$\mathrm{MCO}_3 + 2\,\mathrm{NH}_4\mathrm{ClO}_4 = \mathrm{M}(\mathrm{ClO}_4)_2 + \mathrm{H}_2\mathrm{O} + \mathrm{CO}_2 + 2\,\mathrm{NH}_3$$

REFERENCES:

G. F. Smith and E. G. Koch. Z. anorg. allg. Chem. <u>223</u>, 18 (1935).
G. F. Smith and V. R. Hardy. Z. anorg. allg. Chem. <u>223</u>, 1 (1935).

## Nitrosyl Perchlorate

 $NOClO_4$ 

$$NO_2 + NO + 2 HClO_4 = 2 NOClO_4 + H_2O$$
  
200.9 259.0

An approximately 30% aqueous solution of HClO<sub>4</sub> (100 ml.) is evaporated in a porcelain dish until dense white fumes are evolved  $(142^{\circ}C)$ . The remaining liquid (a mixture of HClO<sub>4</sub> mono- and dihydrates) is poured into a round-bottom flask. A mixture of NO and NO<sub>2</sub> is then introduced (prepared by dropwise addition of 68% nitric acid to NaNO<sub>2</sub>).

From 11 to 16 g. of colorless, thin platelets of  $NOClO_4 \cdot aq$ . form; these are filtered on a Büchner funnel. The yield can be raised almost to theoretical (53 g.) if the filtrate is evaporated and the NO-NO<sub>2</sub> mixture is reintroduced.

After filtration, the crystals are placed in a desiccator and dried for several hours on porous clay over  $P_2O_5$ . It is also a good idea to prefill the desiccator with NO-NO<sub>2</sub> mixture.

After drying the material in a vacuum over  $P_2O_5$  for several days, the water is completely removed and anhydrous NOClO<sub>4</sub> is obtained.

PROPERTIES:

Formula weight  $(NOClO_4)$  129.5. White crystals which decompose with water to form NO, NO<sub>2</sub>, HNO<sub>3</sub> and HClO<sub>4</sub>.

**REFERENCES:** 

- K. A. Hofmann and A. v. Zedtwitz. Ber. dtsch. chem. Ges. <u>42</u>, 2031 (1909).
- K. Kruse, B. Drobny, G. Huck and H. Möller. Z. anorg. allg. Chem. <u>259</u>, 154 (1949).

## Nitryl Perchlorate

## NO<sub>2</sub>ClO<sub>4</sub>

I.

 $\frac{\text{HNO}_3 + 2 \text{ HClO}_4 = \text{H}_3 \text{OClO}_4 + \text{NO}_2 \text{ClO}_4}{63 \quad 201 \quad 118.5 \quad 145.5}$ 

According to Goddard, Hughes and Ingold,  $NO_2ClO_4$  is prepared by reacting anhydrous  $HNO_3$  with anhydrous  $HClO_4$  in a high-vacuum apparatus. The simultaneously formed  $H_3OClO_4$  is reconverted to  $HNO_3$  and  $HClO_4$  by the addition of  $N_2O_5$ . The reaction is carried out in a nitromethane solution, from which the  $NO_2ClO_4$  is obtained by crystallization.

II. According to Gordon and Spinks, dry air  $(0^{\circ}C)$  is passed through a Siemens ozonizer at a rate of 121 iters/hour. The products (ozone and nitrogen-containing gases) are mixed with a much slower stream of chlorine dioxide in an adjacent vessel. White crystals, with the composition NO<sub>2</sub>ClO<sub>4</sub>, are deposited on the walls of the reaction vessel. Goddard, Hughes and Ingold call this compound nitronium perchlorate; Gordon and Spinks call it nitroxyl perchlorate.

PROPERTIES:

At 120°C, the compound decomposes rapidly, but not at an explosive rate.

For other information on the reaction between  $HNO_3$  and  $HClO_4$ , see A. Hantzsch, Ber. dtsch. chem. Ges. 58, 958 (1925).

**REFERENCES:** 

- I. D. R. Goddard, E. D. Hughes and C. K. Ingold. Nature (London) 158, 480 (1946).
- II. W. E. Gordon and J. W. T. Spinks. Canad. J. Res. (B) <u>18</u>, 358 (1940); Chem. Zentr. <u>1942</u> (I), 3180.

## **Periodic Acid**

## H5IO6

Since  $H_5IO_6$  is decomposed catalytically by Pt, electrolysis on Pt cannot be used. The procedure described below, using barium periodate and HNO<sub>3</sub>, makes use of the fact that  $Ba(NO_3)_2$  is insoluble in concentrated HNO<sub>3</sub>, while  $H_5IO_6$  is soluble.

 $\begin{array}{c} \text{Ba}_{3}\text{H}_{4}(\text{IO}_{6})_{2} + 6 \text{ HNO}_{3} = 2 \text{ H}_{5}\text{IO}_{6} + 3 \text{ Ba}(\text{NO}_{3})_{2} \\ 861.9 & 378.1 & 455.9 & 784.1 \end{array}$ 

A 100-g. quantity of  $Ba_3H_4(IO_6)_2$  is moistened with 75 ml. of  $H_2O$  and treated with 200 ml. of colorless nitric acid (d 1.42). The agitated mixture is heated to  $60-70^{\circ}C$  for an hour and is then cooled to  $30-40^{\circ}C$ . The precipitated  $Ba(NO_3)_2$  is filtered off on a glass frit. The residue is washed free of periodate by stirring with concentrated nitric acid. The combined filtrate is evaporated at  $60-70^{\circ}C$  in aspirator vacuum [if more  $Ba(NO_3)_2$  separates, it is filtered and the evaporation continued] until  $H_5IO_6$  begins to precipitate. After cooling, glistening crystals of periodic acid are formed. Since the solution tends to become supersaturated, it is often necessary to wait a long time. The crystals are filtered off and dried in a vacuum at  $50^{\circ}C$ . A second crystal crop may be obtained from the mother liquor by evaporation. The yield is almost quantitative (46 g. vs. the theoretical 52.9 g.).

No rubber tubes or rubber stoppers may be used in this procedure because they would reduce  $HNO_3$  to lower oxides, which in turn would reduce  $H_5IO_6$  to  $HIO_3$ .

PROPERTIES:

Formula weight 227.96. Colorless, hygroscopic crystals which decompose into  $H_2O$ ,  $O_2$  and  $I_2O_5$  at the melting point (130°C).

**REFERENCE:** 

H. H. Willard in: H. S. Booth, Inorg. Syntheses, Vol. I. New York-London, 1939, p. 172.

## **Sodium Periodates**

## Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub>, NaIO<sub>4</sub>

The syntheses of  $Na_3H_2IO_8$  and  $NaIO_4$  use  $NaIO_3$  as the starting material. The latter is either used as such, or in the form of a solution which may easily be prepared from elemental iodine and excess  $NaCIO_3$  in the following manner.

A solution of 125 g. of pure NaClO<sub>3</sub> in 500 ml. of  $H_2O$  is prepared at 45°C in a five-liter flask. The solution is acidified with 2 ml. of concentrated nitric acid. Iodine (100 g.) is then added and an inverted beaker placed over the mouth of the flask to avoid loss of iodine. The reaction mixture is then heated, with constant agitation, to 50-70°C. If the reaction becomes too violent, the flask is cooled by immersion in cold water. The end of the reaction (in about 15 minutes) may be recognized by the disappearance of the iodine color. This solution may be used for the preparation of Na periodate. For each 100 g. of I<sub>2</sub>, 76.9 g. of NaClO<sub>3</sub> is required, and 156.1 g. of NaIO<sub>3</sub> is produced.

## Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub>:

I.

 $\begin{array}{c} NaIO_{3} + 4 NaOH + Cl_{2} = Na_{3}H_{2}IO_{6} + 2 NaCl + H_{2}O\\ 197.9 & 160 & 70.9 & 293.9 & 116.9 & 18 \end{array}$ 

Solid NaOH (140 g.) and, if necessary, another 100 to 200 ml. of  $H_2O$  are added to a beaker containing the iodate solution prepared from 100 g. of  $I_2$  as described above (alternately, 156.1 g. of NaIO<sub>3</sub> may be used). The mixture is vigorously boiled and  $Cl_2$ is introduced as rapidly as possible through a glass tube at least 1 cm. in diameter. This also serves to agitate the reaction mixture continuously and vigorously. After about 10-15 minutes all the alkali is neutralized and no further  $Cl_2$  is absorbed. The solution is then made slightly alkaline with NaOH in order to convert the small amount of the Na<sub>2</sub>H<sub>3</sub>IO<sub>6</sub> byproduct into the less soluble Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub>. On cooling, the precipitate is filtered on a Büchner funnel. The precipitate is washed with cold water and dried at  $110^{\circ}$ C. The yield is about 225 g. (about 97% of the theoretical yield of 231.8 g.).

Other preparative methods:

II. If cylinder chlorine is unavailable, NaIO<sub>3</sub> may be oxidized with  $K_2S_2O_8$  [the use of  $(NH_4)_2S_2O_8$  is not recommended because of poor yields]. Using this method, some sulfate contamination of the product must be expected.

 $\begin{array}{ll} NaIO_3 + K_2S_2O_8 + 4 \ NaOH = Na_3H_2IO_6 + K_2SO_4 + Na_2SO_4 + H_2O\\ 197.9 & 270.3 & 160 & 293.9 & 174.3 & 142 & 18 \end{array}$ 

The NaIO<sub>3</sub> solution prepared as above from 100 g. of I<sub>2</sub> (or a solution of 156.1 g. of NaIO<sub>3</sub>) is gradually reacted with 40 g. of NaOH; it is then diluted with H<sub>2</sub>O to a total volume of 1200 ml. and is then heated to boiling. The stoichiometric quantity (213 g.) of  $K_2S_2O_8$  is then gradually added, followed by 170 g. of NaOH, added in portions. Continuous, vigorous stirring is necessary during the addition; a power agitator is preferable. Following the addition, the mixture is boiled for another 15 minutes, cooled to 40°C, and filtered through a fritted glass filter. A considerable amount of sulfate crystallizes on cooling below 40°C. The Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub> precipitate is washed several times with cold water.

III. The compound may also be prepared from NaI, Br<sub>2</sub> and NaOH:

 $\begin{array}{rrrr} NaI \ + \ 4 \ Br_2 \ + \ 10 \ NaOH \ = \ Na_2 H_2 IO_6 \ + \ 8 \ NaBr \ + \ 4 \ H_2 O \\ 149.9 \ \ 639.3 \ \ 400 \ \ 293.9 \ \ 823.3 \ \ 72 \end{array}$ 

A solution of 50 g. of NaI and 264 g. of NaOH in two liters of  $H_2O$  is prepared in a four-liter breaker. The solution is heated to 80°C and, while mechanically stirred, is gradually reacted with 80 ml. of  $Br_2$  (2 ml./minute). The bromine is added from a dropping funnel with its tube projecting below the surface of the liquid. During the addition the temperature is kept as close as possible to  $80^{\circ}C$ . A precipitate suddenly forms after 30-45 minutes. The drop by drop addition of  $Br_2$  is continued. By a quick movement of the flask, the liquid is decanted from the precipitate and the remainder of the  $Br_2$  is added to the liquid, which is then recombined with the residue. The Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub> is now filtered through a fritted glass filter, washed four times with 25 ml. of H<sub>2</sub>O and air dried. The yield is about 85 g. (87%).

NaIO<sub>4</sub>:

$$\begin{array}{c} Na_{3}H_{2}IO_{6} + 2 HNO_{3} = NaIO_{4} + 2 NaNO_{3} + 2 H_{2}O_{293.9} \\ 293.9 \\ 126 \\ 213.9 \\ 170 \\ 36 \end{array}$$

A 100-g. quantity of  $Na_3H_2IO_e$  is treated with 200 ml. of  $H_2O$  and 55 ml. of concentrated nitric acid (20% excess). If the liquid

is not clear, it is filtered through fritted glass. The filtrate is evaporated until crystals form. It is then cooled to 20°C (cooling to a lower temperature causes  $NaIO_4 \cdot 3H_2O$  to crystallize out) and the precipitate is filtered off, washed with cold H<sub>2</sub>O and dried at 110°C. The yield is about 61 g. (84%). The periodate still contained in the solution may be recovered as the rather insoluble KIO, (about 11 g.) by precipitation with KNO<sub>3</sub>.

PROPERTIES:

White crystals. NaIO<sub>4</sub>: d 3.865.

REFERENCES:

I and II, M. Guichard. Bull. Soc. Chim. France [4] 5, 724 (1909). H. H. Willard in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London, 1939, p. 168.

A. E. Hill. J. Amer. Chem. Soc. 50, 2678 (1928).

E. Müller and W. Jakob. Z. anorg. allg. Chem. 82, 308 (1913).

III. J. Lange and R. R. Paris. J. Pharm. Chim. 21, 403 (1935). P. M. Bernays in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 212.

## **Potassium Periodate**

## **KIO**₄

The preparation is analogous to that of sodium periodate:  $I_2$  is converted to KIO, by means of KCIO, and the KIO, is oxidized with Cl<sub>2</sub>.

The KIO<sub>3</sub> solution obtained from 100 g. of I<sub>2</sub> and 135 g. of KClO<sub>3</sub> (or a solution of 168.6 g. of KIO<sub>3</sub>) is treated with 195 g. of pure KOH (correspondingly more KOH if hydrous), and chlorine is passed through as previously described. The  $K_4I_2O_9$  remains dissolved in this alkaline solution and  $KIO_4$  may be precipitated by making the solution neutral or weakly acidic. The yield is almost quantitative (about 178 g.).

PROPERTIES:

White crystals. d 3.618. Solubility  $(13^{\circ}C)$ : 0.66 g. KIO<sub>4</sub>/100 ml. H<sub>2</sub>O.

REFERENCES:

A. E. Hill. J. Amer. Chem. Soc. 50, 2678 (1928).

H. H. Willard in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London, 1939, p. 171.

## **Barium Periodate**

 $Ba_3H_4(IO_6)_2$ 

I.

 $2 \operatorname{Na_3H_2IO_6}_{587.8} + 3 \operatorname{Ba(NO_3)_2}_{784.1} = \operatorname{Ba_3H_4(IO_6)_2}_{861.9} + 6 \operatorname{NaNO_3}_{510}$ 

The Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub> (about 225 g.) obtained from 100 g. of I<sub>2</sub>, following the directions for Na periodate, is dissolved in one liter of H<sub>2</sub>O and the solution, to which 10 ml. of concentrated nitric acid has been added, is heated to boiling. It is then treated with a hot aqueous solution of 425 g. of Ba(NO<sub>3</sub>)<sub>2</sub>. The mixture is boiled for 1.5 to 2 hours with vigorous stirring, then neutralized with Ba(OH)<sub>2</sub> and left to cool. The barium periodate that crystallizes out is repeatedly washed with hot water and the supernatant liquor decanted. It is finally filtered on a Büchner funnel. The yield is about 330 g.; however, the product still contains some NaNO<sub>3</sub>. II. By starting with KIO<sub>4</sub>, the product may be prepared according to the following equation:

$$\begin{array}{c} 2 \operatorname{KIO}_4 + 3 \operatorname{Ba}(\operatorname{NO}_3)_2 + 4 \operatorname{KOH} = \operatorname{Ba}_3 \operatorname{H}_4(\operatorname{IO}_6)_2 + 6 \operatorname{KNO}_3 \\ 460 & 784.1 & 224.4 & 861.9 & 606.6 \end{array}$$

The procedure is identical to that given above. However, if 100 g. of iodine or 181.2 g. of  $KIO_4$  is used as the starting material, another 88.4 g. of KOH should be added before the introduction of  $Ba(NO_3)_2$ .

**PROPERTIES:** 

Formula weight 861.9. White crystals.

**REFERENCE:** 

H. H. Willard in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London, 1939, p. 171.

## **Chlorine Nitrate**

ClNO<sub>3</sub>

 $\begin{array}{c} Cl_2O + N_2O_5 = 2 \ ClNO_3 \\ 86.9 \ 108.0 \ 194.9 \end{array}$ 

A refrigerated pocket-shaped receiver is used to sublime an excess of  $N_2O_5$  directly onto  $Cl_2O$  in high vacuum. The receiver

containing the  $Cl_2O$  (equivalent to 5 ml. of liquid) is placed in liquid nitrogen so that only the portion holding the  $Cl_2O$  is cooled, and the  $N_2O_5$  thus deposits on the  $Cl_2O$  as a solid. After releasing the vacuum, the receiver is removed from the high-vacuum apparatus, closed off by means of a drying tube, and placed in a refrigerating bath at  $-78^{\circ}C$ . While the bath is slowly warmed to between -20 and  $0^{\circ}C$  (over a period of 15 hours) the components react slowly with each other. The  $ClNO_3$  thus formed is still contaminated with chlorine (an impurity of the  $Cl_2O$ ) and excess  $N_2O_5$ .

In order to eliminate the chlorine, which is difficult to separate, the reaction product is heated to  $30^{\circ}$ C on a water bath and refluxed for one hour. The upper half of the reaction vessel (serving as a condenser) is cooled by a ring-shaped cup filled with Dry Ice. In this procedure the chlorine is volatilized, while the excess  $N_2O_5$  is decomposed to  $NO_2$  and  $O_2$ . The ClNO<sub>3</sub> is then distilled at  $-90^{\circ}$ C in high vacuum, leaving behind the  $NO_2$ .

PROPERTIES:

Colorless to pale yellow, mobile liquid. M.p.  $-107^{\circ}$ C, b.p.  $18^{\circ}$ C (extrapolated).

REFERENCES:

- M. Schmeisser. W. Fink and K. Brändle, Angew. Chem. <u>69</u>, 780 (1957).
- W. Fink. Thesis, Univ. München, 1956.

# Dipyridineiodine (I) Perchlorate

 $[I(C_5H_5N)_2]ClO_4$ 

This compound is prepared in pyridine solution as follows:

$$AgClO_4 + 2C_5H_5N = [Ag(C_5H_5N)_2]ClO_4 [Ag(C_5H_5N)_2]ClO_4 + I_2 = Ag[I + [I(C_5H_5N)_2]ClO_4$$

PROPERTIES:

Colorless, saltlike compound, comparatively stable in air.

**REFERENCE:** 

- H. Carlsohn. Über eine neue Klasse von Verbindungen des positiven einwertigen Jods [A New Class of Univalent Iodine Compounds], Ph. D. Thesis, Leipzig, 1932.
- H. Carlsohn. German Patent 692,324 (1940).

The following compounds may be prepared in a similar way:  $[CIPy_X]NO_3$ ,  $[BrPy_X]NO_3$ ,  $[BrPy_X]CIO_4$ ,  $[IPy_X]NO_3$ ,  $[IPy_X]CIO_4$ (Py =  $C_5H_5N$ ; x = 1 or 2).

- H. Carlsohn, see above; H. Carlsohn. Ber. dtsch. chem. Ges. <u>68</u>, 2209 (1935).
- M. J. Uschakow and W. O. Tschistow. Ber. dtsch. chem. Ges. <u>68</u>, 824 (1935).

 $[IPy_2]F, [BrPy_2]F.$ 

H. Schmidt and H. Meinert. Angew. Chem. 71, 126 (1959).

## Bromine (III) Nitrate

## Br(NO<sub>3</sub>)<sub>3</sub>

# $\begin{array}{l} \text{Br}\text{F}_3 + 3 \text{ N}_2\text{O}_5 = \text{Br}(\text{NO}_3)_3 + 3 \text{ NO}_2\text{F} \\ 136.9 \quad 324.0 \quad 265.9 \quad 195.0 \end{array}$

About 2 to 3 g. of  $BrF_3$  is distilled into a quartz receiver (trap) in high vacuum. About 40 ml. of Freon 11 (CFCl.) is condensed on the BrF<sub>3</sub>, which has deposited on the receiver walls. After releasing the vacuum, the solvent is warmed to about  $-10^{\circ}$ C and the BrF<sub>3</sub> melted into the Freon 11 by hand warming. Part of the BF<sub>3</sub> dissolves, coloring the solution a pale yellow, and part of it forms a fine crystalline suspension. The BF<sub>3</sub>-Freon solution (or suspension) is frozen with liquid nitrogen and a slight excess of finely pulverized N<sub>2</sub>O<sub>5</sub> is added through a tube attached to the receiver and protected against moisture. The liquid nitrogen is replaced by a bath kept at  $-30^{\circ}$ C. This temperature is maintained for several hours, during which it is best to stir the mixture magnetically with a Teflon-coated stirrer. After this, the Freon and NO<sub>2</sub>F are distilled off in high vacuum at -78°C. The NO<sub>2</sub>F can be condensed in a receiver cooled with liquid nitrogen and the Freon in a receiver cooled to  $-140^{\circ}$ C. The excess N<sub>2</sub>O<sub>5</sub> is sublimed off at  $-40^{\circ}$ C in high vacuum, the sublimation taking several hours. Pale yellow Br(NO<sub>3</sub>), remains in vessel.

#### PROPERTIES:

White to pale yellow solid; very sensitive to moisture; slowly decomposes above  $0^{\circ}$ C into Br<sub>2</sub>, O<sub>2</sub> and NO<sub>3</sub>. M.p. 48°C (dec.); soluble in Freon 11 and CCl<sub>4</sub>.

REFERENCE:

M. Schmeisser and L. Taglinger. Angew. Chem. 71, 523 (1959).

# Iodine (III) Nitrate I(NO<sub>3</sub>)<sub>3</sub>

I.

 $ICl_{3} + 3 ClNO_{3} = I(NO_{3})_{3} + 3 Cl_{2}$ 233.3 292.4 312.9 212.8

An excess of ClNO<sub>3</sub> is condensed onto liquid-nitrogen-cooled  $ICl_3$  in high vacuum. After releasing the vacuum the mixture is warmed to 0°C. The reaction starts at -30°C, with evolution of chlorine. The products are allowed to stand for a day at 0°C and the insoluble cake that forms during this time is periodically broken up with a glass rod.

At the end of the 24-hour period, the unreacted  $ClNO_3$  and the traces of  $Cl_2$  are distilled off in high vacuum at  $-70^{\circ}C$ .

II.  $ICl_3 + 3 HNO_3 = I(NO_3)_3 + 3 HCl_{233.3} 189.1 312.9 109.5$ 

Iodine (III) nitrate may also be prepared by treating  $ICl_3$  with anhydrous  $HNO_3$ .

By long and vigorous mixing with a magnetic stirrer,  $ICl_3$  is made into a slurry with Freon. An excess of a solution of anhydrous  $HNO_3$  in Freon is then added drop by drop. A yellow, emulsionlike product is formed which can, to a large degree, be separated as flakes by cooling to  $-78^{\circ}C$ . After settling and decanting the supernatant liquid, all volatile fractions are distilled off at  $-80^{\circ}C$ in high vacuum. In order to remove the  $HNO_3$  completely, the reaction mixture must remain in high vacuum overnight. During this time the temperature of the cold bath should rise to about  $-45^{\circ}C$ . A yellow powder of  $I(NO_3)_3$  remains in the flask.

PROPERTIES:

Yellow, brittle, hygroscopic solid; decomposes above 0°C with softening.

**REFERENCES:** 

I. M. Schmeisser and K. Brändle. Angew. Chem. <u>69</u>, 781 (1957). II. K. Brändle. Thesis, Aachen, 1958.

## Iodine (III) Sulfate

 $I_2(SO_4)_3$ 

Basic and neutral salts are among the compounds containing positive trivalent iodine. The colorless neutral salts are extremely moisture sensitive, while the yellow basic compounds containing the iodosyl group IO<sup>+</sup> are relatively stable.

$$(IO)_2SO_4 \cdot H_2O + 3SO_3 = I_2(SO_4)_3 + H_2SO_4$$
  
.399.8 240 541.8 98

Sulfur trioxide (40-50 g.) is distilled onto 10g. of  $(IO)_2SO_4 \cdot H_2O$ (see below) in a 50-cm.-long and 2-cm. diameter tube. The tube is then sealed off and heated in a paraffin-oil bath until the darkyellow  $(IO)_2SO_4$  converts to a homogeneous, yellow crystalline mass. This requires about 140 hours at 100-120°C. The individual crystals may be readily seen in the tube as long as the latter is still warm and the excess  $SO_3$  is still fluid. After cooling, the product is freed of  $SO_3$  on a porous clay plate placed in a desiccator over  $H_2SO_4$ . The bright yellow  $I_2(SO_4)_3$  crystals are extremely hygroscopic and instantly develop a black color on contact with moist air (separation of  $I_2$ ).

REFERENCE:

F. Fichter and H. Kappeler. Z. anorg. allg. Chem. 91, 134 (1915).

# Iodine (III) Perchlorate

## I(ClO<sub>4</sub>)<sub>3</sub>

 $\begin{array}{rrrr} I_2 \,+\, 6\, HClO_4 \,+\, 3\,O_3 \,=\, 2\, I(ClO_4)_3 \,+\, 3\, H_2O \,+\, 3\,O_2 \\ 253.8 & 602.8 & 144 & 850.6 & 54 & 96 \end{array}$ 

lodine (4 g.) and anhydrous  $\text{HClO}_4$  are precooled separately in ice-salt baths and then mixed. A stream of  $O_2$  containing about 8% ozone is introduced while the mixture is kept at 0°C. The gas stream must be absolutely dry and the reaction flask carefully protected against moisture (be careful in handling  $\text{HClO}_4$ : organic materials must not come in contact with it!). When the solution assumes a transparent greenish color, treatment with  $O_3$  is interrupted and the reaction vessel is left to cool for half an hour in the ice-salt mixture. The greenish-yellow crystals are suctionfiltered on a filter crucible protected from moisture by a CaCl<sub>2</sub> tube and are washed with some cold, anhydrous  $\text{HClO}_4$ .

PROPERTIES:

Extraordinarily moisture sensitive; even at room temperature undergoes internal oxidation. Must therefore be kept cold.

**REFERENCE:** 

F. Fichter and H. Kappeler. Z. anorg. allg. Chem. 91, 134 (1915).

## Iodine (III) Iodate

## $I(IO_3)_3$ or $I_4O_9$

I.

 $\begin{array}{r} 2\,\mathbf{I_2} + 9\,\mathbf{O_3} = \mathbf{I_4O_9} + 9\,\mathbf{O_2} \\ 507.7 \quad 432 \quad 651.7 \quad 288 \end{array}$ 

A stream of  $O_2$  containing about 8% ozone is passed through a U tube, the lower part of which contains iodine. The latter is heated until vaporized. The yellow  $I_4O_9$  product is precipitated in an attached U tube filled with washed and dried glass wool. Contact with moisture must be very carefully avoided during the entire preparation.

A variation of this method, using  $CHCl_3$  solution, was described by Fichter and Rohner.

II. 
$$4 \text{ HIO}_3 (+ \text{H}_3\text{PO}_4) = I_4\text{O}_9 + 2 \text{ H}_2\text{O} + \frac{1}{2}\text{O}_2 (+ \text{H}_3\text{PO}_4)$$
  
703.7 651.7 36 16

Concentrated phosphoric acid (d. 1.7, 20 ml.) is dehydrated by heating in a large Pt crucible. After cooling, 8 g. of powdered  $HIO_3$  is added in portions with stirring and the mixture is carefully heated. Oxygen is evolved, and after about 15 minutes iodine vapor is given off, imparting a yellow color to the reactants. Heating is then interrupted and the crucible is cooled in a vacuum desiccator over  $H_2SO_4$ . A white, pasty mass gradually forms. This is stirred with concentrated  $H_2SO_4$  in order to work it up. When the precipitate has settled, the liquid is decanted, and the crystalline mass is dried on a porous plate in a vacuum desiccator.

Care should be taken to make sure that the substance does not come into contact with moist air during any of the above equations.

**PROPERTIES:** 

Extremely hygroscopic, bright yellow solid; decomposes above  $75^{\circ}$ C with formation of  $I_2O_5$   $I_2$  and  $O_2$ .

**REFERENCES:** 

- R. K. Bahl and J. R. Partington. J. Chem. Soc. (London) <u>1935</u>, 1258.
- M. Beger, Chemiker-Ztg. 33, 1232 (1909).
- F. Fichter and F. Rohner, Ber. dtsch. chem. Ges. <u>42</u>, 4093 (1909).
- F. Fichter and H. Kappeler, Z. anorg. allg. Chem. 91, 142 (1915).

## Oxoiodine (III) Sulfate (IO)<sub>2</sub>SO<sub>4</sub> · H<sub>2</sub>O

I.

 $\begin{array}{c} 2 \text{ HIO}_3 + \text{H}_2\text{SO}_4 = 2 \text{ (IO)}_2\text{SO}_4 \cdot \text{H}_2\text{O} + \text{O}_2 + \text{H}_2\text{O} \\ 351.9 \quad 98 \quad 399.9 \quad 32 \quad 18 \end{array}$ 

A stirred mixture of 6 g. of  $HIO_3$  and 20 g. of concentrated  $H_2SO_4$  is heated in a platinum dish. Oxygen evolves for a few moments and the mixture then assumes a yellowish-brown color. The heating of the mixture is then continued with a smaller flame until violet iodine fumes are observed. As soon as this occurs, heating is stopped and the product is cooled and left to stand for 5-6 days in a desiccator over concentrated  $H_2SO_4$ . The liquid is then decanted from the crust of yellow crystals. The crystals are powdered, washed with a small quantity of the decanted liquid, suction-filtered through fritted glass, and dried on a porous plate in a vacuum desiccator over  $H_2SO_4$ . The yield is 5 g.

II.

$$I_2 + H_2SO_4 + 3O_3 = (IO)_2SO_4 \cdot H_2O + 3O_2$$
  
253.8 98 144 399.8 96

Iodine is dissolved in  $H_2SO_4$  and ozone-containing  $O_2$  is introduced, whereupon a yellow crystalline powder precipitates. It is treated as in method I.

#### **PROPERTIES:**

Yellow, hygroscopic powder, very slightly soluble in cold water. Hydrolysis yields  $I_2$ ,  $HIO_3$  and  $H_2SO_4$ . According to Chrétien, the compound has the formula

According to Chrétien, the compound has the formula  $(IO)_2SO_4 \cdot \frac{1}{2}$  H<sub>2</sub>O; according to the more recent work of Bahl and Partington, the formula is  $(IO)_2SO_4 \cdot H_2O$ . Bahl and Partington consider the product of the reaction of H<sub>2</sub>SO<sub>4</sub> with HIO<sub>3</sub> as a mixture of  $(IO)_2SO_4 \cdot H_2O$  with  $(I_2O_4) \cdot H_2SO_4$ . However, when treated with a small amount of water, both compounds finally end up as  $I_2O_4$ , since the HIO<sub>3</sub> formed in the hydrolysis yields  $I_2O_4$  with  $(IO)_2SO_4 \cdot H_2O$ . Also known as iodosyl sulfate.

**REFERENCES:** 

- P. Chrétien. Compt. Rend. Hebd. Séances Acad. Sci, <u>123</u>, 814 (1896).
- R. K. Bahl and J. R. Partington. J. Chem. Soc. (London) 1935, 1258.
- M. M. P. Muir. J. Chem. Soc. (London) 95, 656 (1909).
- M. Beger. Chemiker-Ztg. 33, 1232, (1909).

# Diiodine Tetroxide $IO \cdot IO_3$ or $I_2O_4$

$$\begin{array}{c} 4 \text{ HIO}_{3} = 2 \text{ I}_{2} \text{O}_{4} + \text{O}_{2} + 2 \text{ H}_{2} \text{O} \\ 703.7 \quad 635.7 \quad 32 \quad 36 \end{array}$$

The tetroxide  $I_2O_4$  is formed through the following intermediate reaction steps:

 $2 \text{ HIO}_3 + \text{H}_2\text{SO}_4 = (\text{IO})_2\text{SO}_4 \cdot \text{H}_2\text{O} + \text{O}_2 + \text{H}_2\text{O}$  $(\text{IO})_2\text{SO}_4 \cdot \text{H}_2\text{O} + 2 \text{ HIO}_3 = 2 \text{ I}_2\text{O}_4 + \text{H}_2\text{SO}_4 + \text{H}_2\text{O},$ 

where the HIO<sub>3</sub> appearing on the left side of equation (2) is formed from  $(IO)_{a}SO_{4} \cdot H_{2}O_{4}$ .

A sample of  $(IO)_2SO_4 \cdot H_2O$  obtained from  $HIO_3$  and  $H_2SO_4$  (see above) is tested to see whether iodine separation takes place when it is shaken with a small amount of  $H_2O$ . If any  $I_2$  separates, the substance is left in the desiccator for a little longer time; if the test shows no iodine, the entire quantity is quickly washed several times with small amounts of water in a fritted glass suction filter until the wash water is free of sulfate. Suction is applied after each washing. The material is then washed with small amounts of absolute alcohol and then with absolute ether. The substance is dried at room temperature on a clay dish placed in a desiccator over calcium oxide.

**PROPERTIES:** 

Formula weight 317.84. Lemon yellow crystals, not hygroscopic. Very slightly soluble in water. Hydrolysis yields  $I_2$  and  $HIO_3$ . At 130°C,  $I_2O_4$  decomposes to  $I_2O_5$  and  $I_2$ . d 4.2.

REFERENCES:

- R. K. Bahl and J. R. Partington. J. Chem. Soc. (London) <u>1935</u>, 1258.
- M. M. P. Muir. J. Chem. Soc. (London) 95, 656 (1909).
- H. Kappeler. Ber. dtsch. chem. Ges. 44, 3496 (1911).

## SECTION 6

Oxygen, Ozone P.W. SCHENK

#### Oxygen

#### 02

Since oxygen commercially available in steel cylinders can be used for most laboratory purposes after a suitable purification by washing with  $\text{KMnO}_4$ , KO H and concentrated  $\text{H}_2\text{SO}_4$ , only two laboratory procedures which yield a particularly pure gas are described. These are an electrolytic procedure and preparation from hydrogen peroxide.

## I. ELECTROLYTIC OXYGEN

Of the numerous devices for the electrolytic preparation of  $O_2$ , the system using pure nickel electrodes in a 30% KOH solution has proven to be particularly effective. Figure 150 shows the apparatus. It consists of a glass cylinder A, about 50 cm. high and 12 cm. in diameter, which is 2/3 filled with 30% KOH. The cathode  $F_2$  is a cylinder of Ni sheet suspended from three Ni wires (or it may consist of a helix made of 2-mm.-thick nickel wire). These three wires, in turn, are attached to a lid D made of Plexiglas or paraffin-treated hardwood. When Plexiglas is used, a conical hole may be drilled and the wire fastened to the lid by means of a matching ground glass stopper. The anode  $F_1$  is placed in a glass bell C, which is attached to the bottom of tube B (diameter about 20) mm.). This tube is inserted through the lid, which is divided in two for this purpose. The halves are then reconnected with two sheet metal disks. The tube may be cemented to the lid with some picein. The anode itself (helical or sheet) is suspended from a Ni wire, which is cemented on top in a constriction E. A drop of white sealing wax is placed on the Ni wire and is pushed into the preheated constriction to form a seal. A cork washer is pushed from above against the sealing wax and the whole assembly is completely sealed with picein. Some glass wool is placed in the bulb of the side arm of tube B to catch the alkali spray. The bulb is connected to a tube filled with palladium asbestos (see section on Palladium

334

Metals for preparation), which is inserted into a small tubular electric furnace. At maximum line voltage, the furnace should



Fig. 150, Electrolytic preparation of oxygen or hydrogen.

reach about  $350^{\circ}$ C and at most  $400^{\circ}$ C. The furnace as well as the adjoining wash tower containing concentrated  $H_2SO_4$  should be fixed to the glass cylinder with collars made of thick Al sheet. This apparatus works best when a uniform stream of very pure  $O_2$  is needed.

If it is desired to know accurately the quantities of products and reactants, a second electrolyzer of the same type is connected in series with the first, and the O<sub>2</sub> evolved in the latter is collected in suitable glass burettes; i.e., the second apparatus is used as a coulometer. Also, any desired mixture with H<sub>2</sub> may be prepared by attaching under the bell a second auxiliary electrode connected in parallel with the cathode through a resistance. The palladium asbestos tube is then omitted, of course. By suitable choice of the resistance, any desired mixture of  $O_2$  and  $H_2$  can be

readily obtained. The auxiliary electrode is insulated by a glass U tube and is introduced into the bell from below.

Given the above dimensions, the apparatus can carry a current of about 5 amp. and furnish about one liter of oxygen per hour. With prolonged high current, the KOH of the electrolyte under the bell is occasionally depleted, resulting in pronounced corrosion of the anode. If the electrolyte becomes dark, it must be remixed, which can be easily done by allowing it to rise in the bell. *Caution: the poles may not be interchanged during the run!* An oxyhydrogen gas explosion, which is then unavoidable, can have very serious consequences because of the concentrated potassium hydroxide solution. For this reason, careful checking of the correct pole connections is indispensable, even during the check-out run.

Other modifications of electrolyzers, which yield very pure  $O_2$ , have been described by Paneth and by Brauer. For the removal of the last traces of finely dispersed droplets from the electrolytically produced gas, see H. Lux.

II. OXYGEN BY CATALYTIC DECOMPOSITION OF HYDROGEN PEROXIDE

An excellent apparatus for the continuous production of fairly large quantities of very pure oxygen has been described by
Von Wartenberg. The half-liter calibrated flask A (Fig. 151) is closed off with a ground glass adapter B which is sealed with ordinary stopcock grease. This adapter is provided with a small

glass winch C, from which a thin spiral of Ni sheet is suspended by a thin Pt wire. The  $6 \times 7$  cm. Ni sheet is platinized in the same way as the electrodes of a conductivity cell and is ignited at dull red heat in a H<sub>2</sub> stream until it becomes light gray. Before the first run, it is allowed to react in 30% H<sub>2</sub>O<sub>2</sub> for some time. This removes the loose Pt particles. Without this treatment, the decomposition of H<sub>2</sub>O would continue even after the cessation of the run and it would also cause premature consumption of the H<sub>2</sub>O<sub>2</sub> in the apparatus. The ground glass adapter also contains a few glass beads which retain the coarse liquid spray. A small pressure release valve containtaining Hg is connected at F. A sixcm. long platinized Cu wire screen is placed in the vertical tube attached

Fig. 151. Preparation of oxygen from hydrogen peroxide.

past stopcock H in order to decompose the last traces of entrained hydrogen peroxide. A small rotameter is attached after the spiral wash flask containing concentrated H<sub>2</sub>SO<sub>4</sub>. If the apparatus is to be operated for a long time, it is placed in a large container full of cold water to remove the heat developed during the run. After a short startup period, the gases dissolved in the liquid are displaced and very pure, H<sub>2</sub>O<sub>2</sub>-free gas is produced. Commerical Perhydrol is used as starting material. With this material, the post evolution of O<sub>2</sub> is reduced in comparison with very pure Perhydrol. The desired rate of evolution is adjusted by increasing or decreasing the submerged surface of the Ni sheet, using the glass winch. The evolution of gas subsides only after the concentration of H<sub>2</sub>O<sub>2</sub> drops to about 1.5%. Half a liter of 30% H<sub>2</sub>O<sub>2</sub> yields about 45 liters of O<sub>2</sub>.

A Kipp generator can also be used for oxygen production. In this case, 3% H<sub>2</sub>O<sub>2</sub> is used and the generator is charged with cubes made of MnO<sub>2</sub> and a binder.

According to Von Wartenberg, very small quantities of  $O_2$  in inert gases can be detected most simply by means of a bead of white phosphorus. The rising vapor streaks of  $P_2O_2$  indicate as little as 0.002 vol. % of  $O_2$ , provided no gases which interfere with the oxidation (H<sub>2</sub>S, SO<sub>2</sub>, halogens, N<sub>2</sub>O<sub>4</sub>, C<sub>2</sub>H<sub>4</sub>, etc.) are present [Z. Elektrochem. <u>36</u>, 296 (1930)]. Von Wartenberg has also described a simple apparatus for detecting  $O_2$  in this manner [Chem. Ing. Techn. <u>26</u>, 418 (1954)]. This apparatus may be used at a moment's notice.

PROPERTIES:

Formula weight 32.000. Colorless, odorless gas. Condenses to a bluish liquid. M.p. -218.4°C, b.p. -183.0°C. Crit. t. -118.8°C. Crit. p. 49.7 atm. d(-183°C) 1.134. Weight per liter (0°C, 760 mm.) 1.429 g.

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#### Ozone

#### $O_3$

The ozonizer (Fig. 152), the principle of which was described by Siemens and Berthelot, is still the most suitable apparatus for the preparation of ozone. Four or six ozone tubes are connected in series and placed together in a glass cylinder filled with dilute  $CuSO_4$  solution, which serves as the external electrode. The glass cylinder is about 30-40 cm. high. The ozone tubes are made of thin-wall soft glass, as uniform as possible. (The author has observed on occasion that Pyrex ozone tubes are not as effective as those of soft glass.) The inner tube has a diameter of 10 to 12 mm. and is fused concentrically to the outer tube. Three small glass beads on the inner tube keep it in the correct position. The positioning of the inner tube by indentations in the outer tube should be avoided. Such indentations decrease the distance between the inner and outer electrode. The consequent higher electric field density at these points results in breakdowns, leading to the destruction of the tube. The distance between the two tubes, i.e., the air gap between them, is about 1 mm. The individual tubes are fused together in pairs, the fusion point being at the bottom. Two pairs are always connected with each other at the top. In this way the O<sub>2</sub> stream enters the first ozone tube at the top and also leaves



Fig. 152. Preparation of ozone (ozonizer). A) flow meter;
B) analysis vessel with decomposition device for ozone;
C) reaction vessel; D) tube with glass fragments and concentrated KOH solution for decomposing excess ozone.

the last tube at the top. The inner tubes are filled with  $CuSO_4$  solution to a level somewhat higher than that in the cylinders. Thin wires of stainless steel 304 are used as electrical leads. Cork stoppers prevent evaporation of the solution. The individual wires are interconnected. The assembly is then connected to one pole of a small transformer W (so-called instrument transformer) which gives about 8000 volts at the secondary. The other pole of the transformer is grounded and is connected with the outer layer—the solution in the cylinder—by an immersed wire. When a 500-cycle alternating current is available, it should be used in preference to the 60-cycle supply since higher ozone concentrations are then obtained. Cylinder  $O_2$  is used as the starting material. It is dried with concentrated  $H_2SO_4$  and its flow controlled by means of flow meter A. If the presence of nitrogen oxides in the product is undesirable, electrolytic oxygen should be used as the raw material.

Determination of the ozone concentration, which is indispensable in many studies, can be carried out by passing the gas through a KI solution containing some solid boric acid. The iodine formed according to the equation:

$$O_3 + 2 KI + H_9O = I_9 + 2 KOH + O_9$$

is titrated with 0.1 N thiosulfate.

A much quicker procedure, which is sufficiently accurate for most purposes, is based on the volume increase associated with the decomposition of ozone according to the equation

$$2O_3 = 3O_2$$

The setup for this determination is shown schematically in Fig. 152 (B).

The two-way stopcock  $H_1$  allows the gas to flow either into analysis vessel B or into reaction vessel C. It is best to lubricate it with  $P_2O_5$ , which is then converted to HPO<sub>3</sub>. To prevent too rapid liquefaction of the acid (which would produce a leak), ordinary stopcock grease is applied at the top and bottom rim of the stopcock. Stopcocks which close at the bottom are also very useful. All other stopcocks in the apparatus can be sealed with ordinary stopcock grease.

The analysis vessel *B* is cylindrical. It contains a thin spiral of Pt wire supported by two sealed-in current leads. The wire glows when the current from a small line transformer is passed through it. The analysis vessel is connected via stopcock  $H_2$  with an  $H_2SO_4$  manometer *M*. Stopcock  $H_3$  is connected to the hood through tube *D*, which will be discussed later. The analysis vessel is placed in a water bath, which must be large enough to prevent temperature fluctuations during the measurement (the bath temperature may be checked with a thermometer).

The run proceeds as follows: The entire apparatus is first flushed with  $O_2$ , stopcock  $H_1$  is turned toward the analysis vessel, and voltage is applied to the ozonizer. (A red control lamp L connected in parallel with the primary of the transformer is highly recommended.) The flow rate is then adjusted by means of flow meter A and the apparatus is flushed for some time. Stopcock  $H_1$  is then turned toward C, and  $H_3$  is closed. After the zero position has been adjusted on the manometer,  $H_2$  is also closed. The heating current for the Pt wire is turned on for a few minutes, and after the heating current has been turned off,  $H_2$  is again opened. The reading is taken when the manometer level ceases to change. The heating process is repeated to make certain that all the ozone has been decomposed. The ozone content of the gas can be determined from the volume increase or pressure increase given by the previously cited equation. The manometer may be calibrated directly in percent ozone. The reliability of the analyzer is checked by the iodometric method. The ozone from the reaction vessel is vented into the hood through tube D, filled with glass fragments. The glass fragments are wetted with concentrated KOH solution, which completely decomposes the ozone.

This device is much safer than the frequently used heated tubes in which the solvent vapors from the reactor may accumulate. If mixed with oxygen, they occasionally ignite and give rise to violent explosions. It is best to place the entire apparatus under a hood, with the transformers and the ozonizer proper being enclosed in separate boxes. At the very least the electrical components should be protected from contact with ozone.

The present apparatus will yield an oxygen stream with an ozone concentration of 10-12%, especially if 500-c.p.s. AC is used. Still higher concentrations can be obtained by cooling the ozonizer. To obtain a constant  $O_3$  content in the gas, Greenwood suggests cooling of the inner tube of the ozonizer to a constant temperature.

**PROPERTIES:** 

Colorless gas, blue in large volumes; characteristic unpleasant odor. Deep blue liquid. Extremely deleterious to the health, particularly in higher concentrations. B.p.  $-112.3^{\circ}$ C, m.p.  $-251^{\circ}$ C. Crit. t.  $-5^{\circ}$ C. d ( $-183^{\circ}$ C) 1.78. Decomposes rubber completely in a few minutes.

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# SECTION 7

Sulfur, Selenium, Tellurium F. FEHÉR

### Sulfur

S

#### PURE SULFUR

Commercial sulfur can be purified to a considerable extent by repeated recrystallization from CS<sub>2</sub>.

A solution of 31.5 g. of S in 70 g. of  $CS_2$  (C.P.) is prepared at room temperature. The solution is filtered and allowed to stand in ice for some time in a closed Erlenmeyer flask. The sulfur precipitate is filtered off and again crystallized several times in the same manner. At the end of this procedure, it is pressed between filter papers to remove all adhering solvent, and is finely pulverized and dried for a few hours at 90 to 100°C. The purified product is bright light-yellow, and retains only a very weak odor of  $CS_2$ .

However, R. F. Bacon and R. Fanelli claim that the material thus purified, as well as all "chemically pure" commercial sulfurs, are still contaminated with organic substances (particularly liquid hydrocarbons) as well as  $H_2S$ ,  $H_2S_X$  and adsorbed gases. The only method capable of removing these stubbornly adhering impurities proceeds as follows:

One kilogram of sulfur is melted in a one-liter, short-neck glass flask on an open flame. During this operation, the flask is covered with an inverted fritted glass crucible. The temperature is slowly increased until the mass begins to foam. As soon as gas evolution subsides, the liquid is heated to boiling, 5 g. of pure MgO is added, and the solution is boiled for 3-4 hours. It is then allowed to stand at  $125^{\circ}$ C overnight. During this time a black sludge settles to the bottom of the flask; this is rapidly separated, together with the MgO, by filtration through a layer of glass wool. The clear filtrate is treated four more times in the same manner, i.e., each time adding 1% MgO, boiling for 25-30 hours, and filtering through glass wool. After a total treatment time of 100-120 hours, the filtrate is very slowly cooled. The freshly purified product is completely gas free. After standing for several days in air, it is again partially contaminated with gases but can be readily degassed by repeated melting and solidifying at a pressure of 1 mm.

The progress of purification can be followed by occasionally testing the condensate collected on the covering frit. Thus, 3 g. of this sulfur is slowly heated to boiling in a Pyrex test tube (95 × 10 mm.) (with particular care being taken that the region of the highviscosity melt is not traversed too quickly), boiled for 2-3 minutes, and allowed to cool and solidify. No black spots should appear at the lower walls of the glass which are in contact with the flame. The test is good to less than  $10^{-4}\%$  impurities. Before the test, the test tube must be rinsed with hot cleaning solution, and during the test it must be protected from dust.

Von Wartenberg describes a simple method for obtaining S with a C content of less than  $10^{-6}$ %. This procedure requires no special attention and takes place over a period of a few days. A quartz tube, electrically heated to 750°C, is suspended in recrystallized boiling S, thus causing CS<sub>2</sub> to form on it. When no further black material precipitates on the tube, the sulfur is distilled in vacuum.

According to Skjerven,  $H_2S$  may be removed from this carbonfree sulfur by treatment with  $SO_2$ .

**PROPERTIES:** 

Very pure sulfur is light yellow and completely odorless; the melt tends to supercool; there is no residue after burning or distilling.

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Abegg, Handbuch der anorganischen Chemie, Vol. IV, 1. p. 142.

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PLASTIC SULFUR

Plastic sulfur is formed by rapid cooling of molten sulfur. Pure S is heated to 250-350°C under a nitrogen blanket. A thin stream of the melt is then poured into a cylindrical vessel (at least 60 cm. high) which has a bottom inlet and top outlet for icewater flow. Strands of plastic sulfur collect in the lower part while the flowers of sulfur forming on the water surface are continuously flushed away.

#### **PROPERTIES:**

Yellow, plastic, viscoelastic, amorphous mass. The strands can be stretched out to many times their initial length, which causes partial crystallization. Repeated stretching produces loss of elasticity. More detailed studies have been made by K. Sakurada and H. Erbring, Kolloid-Z. <u>72</u>, 129 (1935). Insoluble in water, partially soluble in CS<sub>2</sub>.

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COLLOIDAL SULFUR SOLUTION

I. A very stable monodispersed S sol is obtained by mixing acidified  $Na_2S$  and  $Na_2SO_3$  solutions of the proper concentration, followed by peptization.

Solutions of 7.2 g. of C.P. Na<sub>2</sub>SO<sub>3</sub>.7H<sub>2</sub>O and 6.4 g. of C.P. Na<sub>2</sub>S·9H<sub>2</sub>O, each in 50 ml. of distilled H<sub>2</sub>O, are prepared separately. Then 1.5 ml. of the Na<sub>2</sub>SO<sub>3</sub> solution is added with a pipette to the Na<sub>2</sub>S solution. A mixture of 10 ml. of distilled H<sub>2</sub>O and 2.7 g. of concentrated  $H_2SO_4$  is then added in drops with constant stirring up to the point of incipient turbidity (a total of 8 ml. is needed). Then 5.5 g. of concentrated H<sub>2</sub>SO<sub>4</sub> is added to the remaining Na<sub>2</sub>SO<sub>3</sub> solution and the Na<sub>2</sub>S solution is poured in with constant stirring. The mixture is allowed to stand for one hour in an Erlenmeyer flask covered with a watch glass. It is then filtered through a fluted funnel; the precipitate is washed from the outside of the filter with about 100 ml. of H<sub>2</sub>O and is peptized on the filter with 300 ml, of distilled water. About 5 to 10 ml, of the yellowishwhite colloidal sulfur solution running through the filter is poured into 300 ml. of distilled H<sub>2</sub>O, forming a beautiful, reddish opalescent sulfur sol. After 24 hours the slight deposit of solid which may have formed is filtered off. The sol is then stable for weeks. II. More highly concentrated S sols may be obtained by peptization of finely divided S precipitates in the presence of a protective colloid.

After the S precipitate on the fluted funnel, prepared according to method I, has been washed, it is mixed with sufficient 10% agar solution and dilute NaOH to reach a concentration of 0.6% agar and 0.4% NaOH in the sol product, relative to the weight of dry sulfur. The mixture is then decanted from the insoluble components. The resulting sol is very stable. **REFERENCES:** 

- I. A. Janek, Kolloid-Z. 64, 31 (1933).
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# Hydrogen Sulfide

## H<sub>2</sub>S

The hydrogen sulfide gas prepared in the laboratory from FeS and dilute HCl is generally contaminated with acid vapors such as  $H_2$ ,  $CO_2$ ,  $AsH_3$ ,  $N_2$  and  $O_2$ . The HCl can be readily removed by scrubbing with distilled water. To remove  $AsH_3$ , Jacobsen recommends that the gas, predried with  $CaCl_2$ , be passed through a U tube containing dry iodine and glass wool. The As deposits as  $AsI_3$  and the simultaneously formed HI is then scrubbed out with distilled water. After drying with  $P_2O_5$ , the permanent gases  $H_2$ ,  $N_2$  and  $O_2$  can be removed by condensing the  $H_2S$  in a trap cooled with Dry Ice freezing mixture.

A considerably purer gas is obtained by the decomposition of CaS (prepared by the Otto method and pressed into cubes) with very pure dilute hydrochloric acid. The product gas is passed through a wash bottle with distilled water, two wash bottles with KHS solution, a U tube with CaCl<sub>2</sub>, and finally, into a trap cooled with Dry Ice mixture. The  $H_2S$  reevaporated from this trap is contaminated only with a small quantity of CO<sub>2</sub>.

Very pure, air- and  $CO_2$ -free  $H_2S$  can be prepared via the following procedures:

### I. HEATING OF CONCENTRATED Mg(HS)<sub>2</sub> SOLUTION

A one-liter round-bottom flask with a two-hole rubber stopper serves as the gas generator. A separatory funnel (250 ml.) and a gas outlet tube are placed in the stoper. Then 250 ml. of saturated MgCl<sub>2</sub> solution is allowed to flow in from the separatory funnel, followed by the same quantity of H<sub>2</sub>O, and finally, by the same volume of saturated NaHS solution. A uniform H<sub>2</sub>S stream is generated on slight heating of the flask. The gas is free of CO<sub>2</sub>. At the beginning of the run, the apparatus is flushed with nascent H<sub>2</sub>S (use a hood!) until all the air is displaced. After this, the gas is very pure. It is dried over P<sub>2</sub>O<sub>5</sub> and glass wool and can be used for most laboratory purposes. The yield, based on hydrosulfide, is about 80%.

Π.

$$H_2 + S = H_2S$$
  
2.02 32.07 34.09

Especially pure  $H_2S$  can be obtained by synthesis from the elements. The apparatus is shown in Fig. 153. The flask R, made

of high-melting glass, is charged with about 250 g. of carefully purified S (see p. 341). The side arm reaction tube r-b, inclined upward and having an internal diameter of 2.5 cm. and a length of 150 cm., is filled with pea size pumice chips for a length of 80 cm. These are prepurified by boiling with concentrated HC1, dilute  $H_2SO_4$  and then  $H_2O$  until they show no Cl<sup>-</sup> or  $SO_4^{2-}$  reaction. They are then calcined in a nitrogen and then in a hydrogen stream. The part of the tube containing the chips is encased in furnace E and can be heated to  $600^{\circ}C$ . A lead cooling coil is wound around the middle of the tube section that protrudes from the furnace. The tube outlet is closed off with a thick wad of absorbent cotton and a one-hole rubber stopper. From the stopper, a tube leads to the purification and condensation setup (wash bottle 1 contains distilled water; wash bottles 2 and 3: distilled water and glass chips; wash bottle 4: absorbent cotton).



Fig. 153. Preparation of hydrogen sulfide.

At the beginning of the run, carefully purified N<sub>2</sub> is introduced at B until the air is completely displaced from the apparatus. Then  $H_2$  is allowed to flow in at a rate of 8-9 liters/hour, while E is switched on. When the temperature has risen to  $600^{\circ}$ C, A, is cooled with liquid nitrogen and the sulfur in R is heated to boiling. The hydrogen reacts quantitatively to form H<sub>2</sub>S. The heat input at R should be regulated in such a way that no sulfur accumulates in the side arm and a deposit of flowers of sulfur is formed on the cotton wad only after some time has elapsed. A loose condensate of long, shiny  $H_2S$  crystals precipitates in  $A_1$ . It is melted briefly from time to time in order to utilize the condenser volume completely and to avoid plugging of the inlets and outlets. When  $A_1$  is full, stopcock a is opened and  $A_1$  is sealed at c. For purification the material is distilled from  $A_1$  into  $A_2$  in a high vacuum by slowly heating  $A_1$  and cooling  $A_2$  with liquid nitrogen (the intermediate U tubes are cooled with Dry Ice mixture). The first and last portions of the distillate are discarded. About 130 liters of H<sub>2</sub>S

are obtained in 18 hours; the impurity content of the gas is less than 0.01%.

III. Pure  $H_2S$ , completely free of hydrochloric acid, is obtained in a simple manner by decomposition of a concentrated aqueous solution of C.P. Na<sub>2</sub>S·9H<sub>2</sub>O with 20-30% phosphoric acid (C.P.). The acid is allowed to drip slowly from a separatory funnel into the sodium sulfide solution, and the evolving gas is dried over CaCl<sub>2</sub> and P<sub>2</sub>O<sub>5</sub>.

Hydrogen sulfide is stored in liquid form at low temperature or in gasometers over saturated NaCl solution. Mercury may also be used as sealing fluid when the gas is completely pure and dry.

### PROPERTIES:

Colorless gas with an unpleasant odor reminiscent of rotten eggs; highly toxic. Reducing agent for many substances; for this reason  $H_2S$  cannot be dried with concentrated  $H_2SO_4$ .

M.p.  $-83^{\circ}$ C, b.p.  $-59^{\circ}$ C, crit. t.  $+100.5^{\circ}$ C, crit. p. 98.0 atm. gage; d (b.p.) 0.9504; wt. per liter 1.5392 g. Solubility at 760 mm. in 1 volume of H<sub>2</sub>O (0°C): 4.67; (20°C): 2.58; (100°C): 0.81 volume of H<sub>2</sub>S; for further details, see D'Ans-Lax (1949), Table 332632 A, 7, p. 970; also soluble in ethanol.

In crystalline  $H_2S$ , the S atoms form a face-centered cubic lattice; probably type C1 or C2.

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# **Crude Sulfane**

 $H_2S_x$ 

I.

$$Na_{2}S + yS = Na_{2}S_{x} (x = y + 1)$$
$$Na_{2}S_{x} + 2 HCl = H_{2}S_{x} + 2 NaCl$$

An aqueous solution of sodium polysulfide with the approximate composition  $Na_2S_{5.5}$  serves as the starting material. It is

prepared as follows: A two-liter, round-bottom, long-neck flask is placed on a steam bath and filled with 500 g. of  $Na_2S \cdot 9H_2O$ , which is then heated with 250 g. of powdered S for three hours. Frequent shaking is necessary. The sulfide melts in its water of crystallization and dissolves most of the sulfur, with the color of the mixture becoming deeper. The cooled, dark red-brown solution is diluted with 400 ml. of  $H_2O$ . It is then rapidly suction-filtered to remove S residue and other impurities, and the filtrate is diluted with water to one liter.

Next, a five-liter filtration jar is thoroughly cooled on the outside with an ice-salt mixture, and 2 kg. of finely crushed ice and two liters of pure concentrated hydrochloric acid (d 1.19) are added one after the other. When the acid reaches a temperature of -15 to  $-20^{\circ}$ C, the addition of Na<sub>2</sub>S<sub>x</sub> solution is begun. The solution (in a separatory funnel) is added with thorough stirring over a period of 1-1.5 hours. The solution becomes cloudy during the reaction because of the formation of a milky-white sulfur emulsion. The  $H_2S_X$  product settles at the bottom as a yellow oil. The addition rate should be so controlled that the temperature does not rise above  $-10^{\circ}$ C, if at all possible, and in no case above  $-5^{\circ}$ C; otherwise, decomposition of the  $H_2S_x$  and evolution of  $H_2S$  occurs. After all the  $Na_2S_X$  solution has been added and the product oil has completely settled, the mother liquor is decanted and the oil is washed in the filtration jar with 1 N HCl and decanted in a separatory funnel. A small quantity of P<sub>2</sub>O<sub>5</sub> is added to the crude sulfane. The product has the approximate composition  $H_2S_{5,5}$ ; it can be stored in a closed glass vessel at 0°C for a fairly long period without change. Before further use, the  $P_2O_5$  is filtered off through glass wool. Freshly prepared  $H_2S_X$  should form a clear solution with pure benzene. The yield is 260 g. (160 ml.) of oil, i.e., 87% relative to the starting sulfur.

If crude sulfane is to be prepared continuously in fairly large quantities, the apparatus described by F. Fehér and W. Laue should be used.

Use of sulfane for the preparation of  $H_2S_2$  and  $H_2S_3$ : A crude sulfane which is especially rich in  $H_2S_4$  and is therefore suited for the production of tetrasulfane is prepared in the following manner:

A solution of  $Na_2S_2$  is prepared by heating 480 g. of  $Na_2S \cdot 9H_2O$ , 64 g. of powdered S and 500 ml. of  $H_2O$  for three hours in a twoliter round-bottom flask on a steam bath. The cooled solution, suction-filtered to remove S residues, is then decomposed in the manner described above, using four liters of dilute HCl. The yield is 32 ml. of crude sulfane having the approximate composition  $H_2S_{4.5}$ .

II. In general, excess sulfane,  $H_2S_n$ , which also acts as a solvent, reacts with a chlorosulfane  $S_mCl_2$ , producing quantitative HCl

separation and chain increase and forming a higher sulfane  $H_2S_{2n}+m$  according to the following scheme:

$$H - S_{n} - [H + Cl] - S_{m} - [Cl + H] - S_{n} - H = H_{2}S_{2n+m} + 2 HCl$$
  
Excess  $H_{2}S_{n}$ 

The  $H_2S_{2n+m}$ , which is formed as an intermediate, reacts in turn with the  $S_mCl_2$ . Again, HCl is split off. The formation of a homologous mixture of sulfanes (= crude sulfane,  $H_2S_X$ ) is based on this fact. (See also p. 353).

The number of moles of reagents, a and b, in the equation below gives the average composition x and the quantity c of crude sulfane formed when the quantity d of unreacted sulfane distilled off after the reaction is determined. The equation was derived on the basis of assumption about the molecular distribution obtained in the condensation of sulfanes and halosulfanes (F. Fehér and W. Laue); however, it holds only for the above described reactions under the further restriction that no other reactions take place.

$$\begin{aligned} \mathbf{a}\mathbf{H}_{\mathbf{2}}\mathbf{S}_{n} &+ \mathbf{b}\mathbf{S}_{m} \ \mathbf{Cl}_{\mathbf{2}} = \mathbf{c}\mathbf{H}_{\mathbf{2}}\mathbf{S}_{x} + 2 \ \mathbf{b}\mathbf{H}\mathbf{Cl} + \mathbf{d}\mathbf{H}_{\mathbf{2}}\mathbf{S}_{n} \\ \mathbf{x} &= \frac{\mathbf{a}\mathbf{m} + \mathbf{n}\left(2\mathbf{a} - \mathbf{b}\right)}{\mathbf{a} - \mathbf{b}}; \ \mathbf{c} = \frac{\mathbf{b}\left(\mathbf{n} + \mathbf{m}\right)}{\mathbf{x} - \mathbf{n}}; \ \mathbf{d} = \mathbf{a}\left(1 - \frac{\mathbf{b}}{\mathbf{a}}\right)^{2} \end{aligned}$$

Interfering side reactions which lead to a higher S content of the crude sulfane are favored, among other things, by too violent a reaction and by too small a sulfane excess. Because of its ease of preparation, only  $H_2S$  (n = 1) is of importance as a raw material for "crude oil." The reagents for the preparation of chlorosulfane are  $SCl_2$  (m = 1) and  $S_2Cl_2$  (m = 2). Elementary chlorine, which reacts very violently, may also be used. Since the product is either distilled or cracked, the formation of side products is of no particular importance.

For example, the reaction of a 7-mole excess of  $H_2S$  with  $SCl_2$  yields a crude oil with the approximate composition  $H_2S_{4.5}$  and containing fairly large quantities of  $H_2S_3$ . With  $S_2Cl_2$ , a crude oil of the approximate composition  $H_2S_{5.2}$ , containing a large amount of  $H_2S_4$ , is obtained. These reactions are carried out in the same apparatus and in a manner analogous to that described for the preparation of tetrasulfane (p. 354). The reaction of liquid  $H_2S$  with  $Cl_2$  yields mainly  $H_2S_3$  and  $H_2S_4$ , plus a small amount of  $H_2S_2$ .

The choice of the second component in the reaction with liquid  $H_2S$  usually depends either on the sulfane which is to be distilled directly from the crude mixture or, if cracking is to follow the primary reaction, on the composition of the sulfane mixture which is best suited for that process.

#### SYNONYMS:

"Crude perhydrogen sulfide," "crude hydrogen persulfide," "hydrogen polysulfide," or just "crude oil." For the nomenclature of chain compounds of sulfur, see F. Fehér and W. Laue, Z. Naturforsch. 8b, 11 (1953).

#### **PROPERTIES:**

Yellow liquid resembling olive oil, pungent odor of camphor and  $S_2Cl_2$ .

No definite m.p.; solidifies in liquid air to form a glassy mass which softens again on heating over a fairly broad temperature range. On heating, cracking to lower hydrogen sulfides (see  $H_2S_2$  and  $H_2S_3$ ) takes place. Gradually decomposed by water with evolution of  $H_2S$ . Spontaneous decomposition is caused by alkalis and sometimes even by the alkali content of glass. For this reason, all vessels which come in contact with the product must be "neutralized" beforehand by rinsing with hot, concentrated hydrochloric acid or by treatment with gaseous HCl. Soluble in  $CS_2$ ,  $C_6H_6$  and CHCl<sub>3</sub>.

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- II. F. Fehér, W. Laue and J. Kraemer, Z. anorg. allg. Chem. 281, 151 (1955);
   F. Fehér and W. Laue, Z. anorg. allg. Chem. 287, 45 (1956);
   F. Fehér and W. Kruse, Z. anorg. allg. Chem. 293, 302 (1957).

### **Pure Sulfanes**

Since sulfanes are readily decomposed by alkali, metals, dust, cork, rough surfaces and moisture, all glass vessels which come in contact with them (even glass wool) must be thoroughly cleaned, carefully dried and stored in a dust-free atmosphere.

Glass vessels are rinsed with hot cleaning solution, then with hot, pure, concentrated hydrochloric acid, and finally with distilled water. Apparatus contaminated with sulfur can be precleaned with  $CS_2$  or hot ammonium sulfide solution.

Ground glass connections should be sealed with silicone grease. No  $NH_3$  gas (e.g., from bottles of concentrated ammonia or ammonium sulfide in the area) may be present in the atmosphere. All reactions with sulfanes must be protected from moisture, and pure starting materials must be used.

These precautions must always be very carefully adhered to when working with sulfanes, even when this is not expressly pointed out in the preparation.

### I. DISULFANE AND TRISULFANE, H<sub>2</sub>S<sub>2</sub>, H<sub>2</sub>S<sub>3</sub>

Both of these hydrogen sulfide compounds are formed by cracking of crude sulfane, using the apparatus shown in Fig. 154.

$$H_2S_x = H_2S_2 + (x-2)S$$
  
 $H_3S_x = H_3S_3 + (x-3)S$ 

Since the yields of H<sub>2</sub>S<sub>2</sub> and H<sub>2</sub>S<sub>3</sub> improve when only small quantities of  $H_2S_x$  are cracked at a time, the process is carried out as follows. First, flask a (300 ml.) is heated in a paraffin bath to 110°C while the apparatus is evacuated to 12-15 mm. Then 15 ml. of  $H_2S_x$  (see p. 346 for preparation) is introduced through d. After a short time the condenser surface is coated with fine droplets. The H<sub>2</sub>S<sub>3</sub> collects slowly but at a uniform rate in receiver b, which is at room temperature, while the H<sub>2</sub>S<sub>2</sub> is condensed in trap c, which is cooled with Dry Ice-methanol mixture. Then the bath temperature at a is slowly increased to  $125^{\circ}C$  over a period of 20 minutes. The flask contents are cooled to 110°C and 15 ml. of  $H_2S_X$  is again added through d. The 20-minute heating procedure is repeated. After two portions of H<sub>2</sub>S<sub>X</sub> have been cracked, the vacuum is released and air is slowly introduced into the apparatus, passing through the CaCl<sub>2</sub> and soda-lime tubes. The ground glass joint at a is then quickly disconnected, and the hot, still liquid



Fig. 154. Preparation of di- and trisulfane.

350

residue is poured out, whereupon it solidifies. After the ground glass joint has been resealed, the cracking is continued and the residue is again removed after two 15-ml. additions of  $H_2S_x$ . About 25 ml. of  $H_2S_3$  and 15 ml. of  $H_2S_2$  are obtained from 120 ml. of freshly prepared  $H_2S_x$ . With aged  $H_2S_x$ , the yield of  $H_2S_2$  increases, while that of  $H_2S_3$  is reduced. Both products are nearly pure; at most, each is contaminated by a small amount of the other.

Use of the Bloch and Höhn suction device to remove the cracking residue is not recommended since the residue solidifies easily in the narrow tubes and is then difficult to remove.



Fig. 155. Distillation of disulfane.

The disulfane is further purified by distillation in the apparatus shown in Fig. 155.

The low-temperature column k is a Liebig condenser filled with glass Raschig rings. It is cooled by circulating methanol from a constant temperature bath t. The distillation flask d is placed in a water bath w at room temperature. The trap f for collecting the pure disulfane is attached to k with an adapter. The liquidnitrogen-cooled double trap l is connected to a vacuum pump.

After the crude disulfane, which has been dried with  $P_2O_5$  and filtered through glass wool, is charged into the distillation flask d, the column is cooled to  $-15^{\circ}C$  and the system is slowly evacuated to 15 mm. The flask contents foam vigorously, releasing the dissolved  $H_2S$ . When the initial foaming subsides, condensation of the pure disulfane is begun by cooling f with a Dry Ice-methanol mixture. Further distillation does not require any special attention.

Purification of 100 g. of disulfane takes about 1.5 hours.

The trisulfane is further purified by distillation in the apparatus shown in Fig. 156.

The 500-ml. flask d, immersed in water bath w, carries a Liebig condenser  $k_1$  filled with glass Raschig rings and a spiral



condenser  $k_{2}$ . The latter expands at the bottom to a toroid, from which an outlet leads to receiver v. A horizontal connecting tube leads from  $k_2$  to trap f. The latter is followed by double trap l, cooled with liquid nitrogen, and this in turn is connected to a high-vacuum pump. Each condenser is provided with its own constant-temperature circulation system. The run is started by adjusting the temperature in  $k_1$  to 42°C. Then flask d is filled with the crude trisulfane and attached to the apparatus. Trap f is cooled with Dry Ice-methanol and the system is slowly evacuated. When a pressure of 1.5 mm, is reached,  $k_2$  and v are cooled with ice water and water bath w is heated to 52°C. The distillation now proceeds without further attention;  $H_2S_3$  condenses in v. The more volatile oils condense in f and the less volatile components in  $k_1$ , from which they flow back into flask d. The temperatures in  $k_1$ ,  $k_2$  and w must be constant during the run. The pure distillate generally has the composition  $H_2S_{3,00}$  - 3.03.

Equipment for the continuous preparation of larger quantities of  $H_2S_2$  and  $H_2S_3$  is described by F. Fehér and M. Baudler, Z. anorg. Chem. 253, 170 (1947); 254, 251 (1947); and F. Fehér, W. Laue and G. Winkhaus, Z. anorg. allg. Chem. 288, 113 (1956). This

equipment may also be used to prepare disulfane in yields greater than those described above.

### II. TETRASULFANE, H<sub>2</sub>S<sub>4</sub>

Tetrasulfane is obtained from crude sulfane in the apparatus shown in Fig. 157.

The modified Utzinger tube h, the lower part of which is flattened to increase the vaporization surface as much as possible (see cross section), is surrounded by a heating jacket and carries a separatory funnel t and a 500-ml. flask r. The attached condenser  $k_1$  and the spiral condenser  $k_2$  (which is identical to that used in trisulfane purification and has a toroidal expansion and an outlet to the receiver v) are provided with separate constanttemperature circulating systems. Again, condenser  $k_2$  is connected to trap f, which is immersed in a Dry Ice-methanol mixture. Trap f adjoins double trap l, cooled with liquid nitrogen, which is connected to a high-vacuum pump. The connecting sections of the condensers are covered with asbestos insulation  $t_1$  and  $t_2$ .

The system is evacuated until a pressure of  $10^{-3}$  mm, is attained. Tube h is heated to 75°C,  $k_1$  to 50°C, and  $k_2$  and the water bath to 15°C. Then crude oil of composition  $H_2S_{4.5-5.0}$  (preparation on p. 346) is allowed to drip slowly from the separating funnel t into the tube h. The volatile components of the oil distill off while the residue collects in r. Condenser  $k_2$  separates the tetrasulfane from the lighter and heavier components of the distillate. The product is collected in v. An  $H_2S_2-H_2S_3$  mixture collects in f while the higher-boiling fractions flow from  $k_1$  into r.

III. TETRA-, PENTA-, HEXA-, HEPTA- and OCTASULFANES  $H_2S_4,\,H_2S_5,\,H_2S_6,\,H_2S_7,\,H_2S_8$ 

$$(a-2) H_2S_n + H_S_n - H + Cl - S_m - Cl + H - S_n - H = H_2S_{2n+m} + 2 HCl + (a-2) H_2S_n$$
  
 $n = 1, 2, m = 1, 2, 3, 4, 5, 6.$ 

In general, excess sulfane  $H_2S_n$  reacts with a chlorosulfane  $S_mCl_2$ , splitting off HCl quantitatively and polymerizing to form a homologous mixture of chain sulfanes. The formation of such a mixture is predicated on the fact that the product of the primary reaction reacts further with the chlorosulfane (compare p. 348).

If, however, a very large excess of the  $H_2S_n$  component (a) is used, individual sulfanes can be obtained via direct synthesis. In this case, the rate of formation of the sulfane is so much larger than that of the homologs that it is reaction-controlling. The excess of the sulfane,  $(a - 2)H_2S_n$ , is distilled off after the reaction. Using this principle, it is possible to obtain, with appropriate chlorosulfanes, almost pure  $H_2S_4$ ,  $H_2S_5$ ,  $H_2S_6$ ,  $H_2S_7$  and  $H_2S_8$ . The preparative procedures for all these products are analogous, the reaction conditions mild, and the products almost uncontaminated with each other.

These sulfanes are generally best prepared from the following reactants:

Desired sulfane	Sulfane compo	Sulfane component	
H₂S₄	H <sub>2</sub> S		S <sub>2</sub> Cl <sub>2</sub>
H <sub>2</sub> S <sub>5</sub>	H <sub>2</sub> S <sub>2</sub>		SČ1 <sub>2</sub>
~ •	or $H_{2}S$		S <sub>3</sub> Cl <sub>2</sub>
H <sub>2</sub> S <sub>6</sub>	H <sub>2</sub> S <sub>2</sub>		S <sub>2</sub> Cl <sub>2</sub>
$H_2S_7$	H <sub>2</sub> S		$S_{5}Cl_{2}$
	or $H_{2}S_{2}$		S <sub>3</sub> Cl <sub>2</sub>
H <sub>2</sub> S <sub>8</sub>	H <sub>2</sub> S <sub>2</sub>		S <sub>4</sub> Cl <sub>2</sub>

As an example of the procedure and the ratios of the reactants used, the preparation of tetra- and pentasulfane are described.

### a) TETRASULFANE, H<sub>2</sub>S<sub>4</sub>

The reaction is carried out in a 500-ml., four-neck flask, provided with one male and three female ground glass connections for  $P_2O_5$  drying tube, separating funnel, low-temperature thermometer, and a mercury seal stirrer. The product is poured out through the neck with the male connection.

To prepare the  $H_2S$ , a concentrated aqueous solution of C.P.  $Na_2S \cdot 9H_2O$  (1800 g.) is decomposed with C.P. 25%  $H_3PO_4$  (3400 g.) (compare p. 346) and the gas is passed through CaCl<sub>2</sub> and  $P_2O_5$  drying tubes into a cooled trap (methanol-Dry Ice) where about 150 ml. of  $H_2S$  is condensed. From this trap, the  $H_2S$  is distilled into the reaction flask, passing through the CaCl<sub>2</sub> and  $P_2O_5$  drying tubes. The reaction flask is immersed in a Dewar flask containing Dry Ice-methanol mixture and the gas is thus liquefied.

About 20 g. of dichlorodisulfane (about 0.15 mole) is carefully added from a separatory funnel. The addition temperature is  $-80^{\circ}$ C, stirring is employed, and the dichlorosulfane is added drop-bydrop over a period of two hours. It is advisable to precool the chlorosulfane. Thus, a bent capillary is attached to the outlet tube of the separatory funnel. This capillary is immersed in the liquid H<sub>2</sub>S and then discharges above the surface of the H<sub>2</sub>S. The reaction is complete after about 12 hours, when the reaction mixture has become colorless. The excess  $H_2S$ , contaminated with HCl, is slowly evaporated by removing the Dewar flask. The last traces of  $H_2S$  are removed by briefly applying an aspirator vacuum. Clear, almost pure tetrasulfane remains in the flask. Since  $H_2S$  is highly toxic, the preparation must be carried out under a good hood.

b) PENTASULFANE, H<sub>2</sub>S<sub>5</sub>

The reaction flask (described in the preparation of tetrasulfane; see above) containing 100 g.  $H_2S_2$  is cooled with Dry Ice-methanol to -60 to -65°C in a Dewar flask. From the separatory funnel, 10 g. of SCl<sub>2</sub> (about 0.1 mole) is allowed to drip in slowly (with stirring) in such a way that the temperature of the mixture does not rise above  $-55^{\circ}$ C. In contrast to the preparation of tetrasulfane, the use of a bent outlet capillary is not recommended, since it plugs easily. The Dewar flask is removed 15 minutes after the completion of the dropwise addition. When the mixture has warmed to 0°C, the temperature is held constant until the evolution of HCl has, for all practical purposes, ended; the mixture is then stirred for another 0.5 hour at room temperature. The total reaction time is 1.5 hours.

The clear mixture is poured into a 250-ml., two-neck flask. One neck of this flask, the one connected to a receiving trap, is shaped as shown in Fig. 158. This intricate shape is necessary since the  $H_2S_2$  to be evaporated tends to superheat and splash.



Fig. 158. Evaporation of excess disulfane.

The other neck is a male ground glass piece and is used for pouring out the product. Most of the  $H_2S_2$  is removed in a vacuum of 15 mm., the remainder in high vacuum (about  $10^{-3}$  to  $10^{-4}$  mm.). The product is also briefly heated to 30 to  $40^{\circ}$ C during the last stage of  $H_2S_2$  removal. The generally quite clear greenish-yellow to yellow residue represents the desired product, which is quite pure.

**PROPERTIES:** 

### H<sub>2</sub>S<sub>2</sub>:

Formula weight 66.15. M.p.  $-89.6^{\circ}$ C, b.p. 70.7°C; d. (20°C) 1.334. Light yellow liquid.

H<sub>2</sub>S<sub>3</sub>:

Formula weight 98.21. M.p. -52 to  $-54^{\circ}$ C; d (20°C) 1.491. On heating, cracking to H<sub>2</sub>S<sub>2</sub>, H<sub>2</sub>S<sub>2</sub> and S occurs. Light yellow liquid of somewhat deeper color than H<sub>2</sub>S<sub>2</sub>. Both H<sub>2</sub>S<sub>2</sub> and H<sub>2</sub>S<sub>3</sub> have a pungent odor of camphor and S<sub>2</sub>Cl<sub>2</sub>; the vapors are irritating to the eyes and mucous membranes. On prolonged standing, both are converted to homologs richer in sulfur, giving off H<sub>2</sub>S. With water they decompose with evolution of H<sub>2</sub>S and precipitation of S. Soluble in CS<sub>2</sub>, C<sub>5</sub>H<sub>5</sub> and CHCl<sub>3</sub>.

H₂S₄:

Formula weight 130.28. Bright light-yellow liquid similar to olive oil, pungent odor. Somewhat less easily decomposed than  $H_2S_2$  and  $H_2S_3$ ; like these, it is converted on prolonged standing into homologs richer in sulfur. No definite melting point; solidifies at about  $-85^{\circ}C$  to a white, glassy mass which gradually softens over a fairly wide temperature range when heated. Cracked by heating to lower hydrogen sulfides, particularly  $H_2S$ . d (20°C) 1.582.

H<sub>2</sub>S<sub>5</sub>:

Formula weight 162.35. d (20°C) 1.644.

H<sub>2</sub>S<sub>6</sub>:

Formula weight 194.41. d (20°C) 1.688.

H<sub>2</sub>S<sub>7</sub>:

Formula weight 226.48. d (20°C) 1.721.

H<sub>2</sub>S<sub>8</sub>:

Formula weight 258.54. d (20°C) 1.747.

The properties of the higher sulfanes (from  $H_2S_5$  up) are similar to those of the lower homologs. They are greenish-yellow liquids, changing to a deeper yellow with increasing chain length. At 0°C, the substances are stable for several days if they are protected from contact with air.

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# Ammonium Hydrogen Sulfide NHLHS

 $NH_3 + H_2S = NH_4HS$ 17.03 34.08 51.11

Carefully dried  $NH_3$  and  $H_2S$  are alternately added to an Erlenmeyer flask containing anhydrous ether at 0°C; the flask must be carefully protected from moisture. White crystals of pure  $NH_4HS$ precipitate. These readily decompose again at room temperature into  $NH_3$  and  $H_2S$  and must therefore be quickly suction-filtered and freed of adhering ether by pressing between filter papers. When sealed in glass ampoules,  $NH_4HS$  is stable for some time.

**PROPERTIES:** 

White, needle-shaped crystals; easily decomposed; dissociation pressure at room temperature about 350 mm. d 1.17.

Readily soluble in water and ethanol, forming colorless solutions which rapidly yellow in air; insoluble in ether and benzene.

Tetragonal crystals, space group  $D_{4h}^7$ .

**REFERENCE:** 

J. S. Thomas and R. W. Riding, J. Chem. Soc. (London) <u>123</u>, 1181 (1923).

# Sodium Hydrogen Sulfide NaHS

 $\begin{array}{c} C_2 H_5 ONa + H_2 S = N_a HS + C_2 H_5 OH \\ 68.06 & 34.08 & 56.07 & 46.07 \end{array}$ 

Carefully dehydrated  $C_2H_5OH$  (20 ml.) and small, bright pieces of Na (2 g.) are placed in a 150-ml. glass flask equipped with a reflux condenser and an adapter, closed at first, for introducing a gas

inlet tube. A drying tube on the reflux condenser prevents penetration of atmospheric moisture. When the ethoxide begins to separate, just enough additional C<sub>2</sub>H<sub>5</sub>OH is added to keep all solute in solution at room temperature. A total of about 40 ml. of C<sub>2</sub>H<sub>E</sub>OH is needed. Then a glass tube, reaching to the bottom of the flask, is introduced through the adapter and sealed in tightly with a rubber stopper or a piece of hose. A fast stream of pure, carefully dried H<sub>2</sub>S (see p. 344 ff.) is introduced through this tube into the solution. After a few minutes, an abundant, fine crystalline precipitate of slightly impure NaHS separates out. The solution is saturated with H<sub>2</sub>S and quickly suction-filtered. The filtrate is transferred into a dry Erlenmeyer flask and 50 ml. of pure absolute ether is added. A dense, pure white precipitate of NaHS separates immediately. It is allowed to settle in the stoppered flask, and small portions of ether are added as long as crystals continue to form in the supernatant liquid. In all, about 110 ml, of ether is required for the precipitation. The precipitate is quickly suction-filtered, washed with absolute ether, and transferred to a vacuum desiccator. The yield is 4.3 g. of NaHS. The product is almost analytically pure, Redissolution in absolute C<sub>2</sub>H<sub>z</sub>OH and reprecipitation with ether readily yields a completely pure substance.

PROPERTIES:

White, granular crystalline powder; very hygroscopic. Turns yellow on heating in dry air, orange at higher temperatures; melts to a black liquid at about  $350^{\circ}$ C. d 1.79.

Very soluble in water; moderately soluble in  $C_2H_5OH$ . Pure NaHS dissolves in hydrochloric acid with vigorous evolution of  $H_2S$ to form a clear solution. Aged products give a weakly yellowish, opalescent solution.

Rhombohedrally distorted NaCl structure, space group  $D_{3d}^{5}$ .

**REFERENCES:** 

A. Rule, J. Chem. Soc. (London) <u>99</u>, 558 (1911).

W. Teichert and W. Klemm, Z. anorg. allg. Chem. 243, 86 (1939).

# Sodium Sulfide

 $Na_2S$ 

I.

 $2 Na + S = Na_2S$ 45.99 32.07 78.06

The reaction is carried out in liquid ammonia using the apparatus of Fig. 159.



Fig. 159. Preparation of sodium sulfide.

A weighed tube containing very pure, vacuum distilled Na (see section on Alkali Metals) is opened by breaking off the tip and is quickly pushed into the side arm A of the apparatus. The latter is immediately capped off. After evacuating through B, A is carefully heated until the metal melts and flows through the capillary k onto the frit plate F. The small amount of oxide that formed when the tip was broken off remains in the ampoule. A vigorous stream of dry, carefully purified N<sub>2</sub> is then introduced through B, A is opened again, the empty ampoule is removed, and the weight of reactant Na is determined by reweighing the ampoule. The quantity of sulfur required for the reaction is introduced at C. Vessel R is cooled with a Dry Ice bath, and about 100 ml. of pure NH<sub>3</sub> is condensed on top of the sulfur. The frit plate F and the sharply bent part of tube u are then cooled until some NH<sub>a</sub> collects above the frit; a portion of the Na dissolves in this. The ammonia vapor then produces a higher pressure in the space above F than in R, and this ammoniacal Na solution is forced into R where it contacts the ammoniacal S solution. This procedure is repeated until all the Na has been extracted from the frit with small portions of NH<sub>3</sub>. At first, sodium polysulfides are formed in R because of the excess S; these gradually degrade to Na<sub>2</sub>S as more Na solution is added. By gradually removing the cooling bath at R, the NH<sub>a</sub> is then allowed to evaporate slowly over a period of several hours. Toward the end of this period, a white precipitate separates. The last traces of NH<sub>a</sub> must be evaporated particularly carefully to prevent spattering of the crystals. Finally, the apparatus is evacuated for some time with an oil pump and vessel R is heated for 0.5 hour at 400-500°C. The crystals are pulverized by vigorous shaking and are forced into storage tubes K by tipping the apparatus. The tubes are sealed off in vacuum or N<sub>2</sub> atmosphere. The Na<sub>2</sub>S is very pure; the yield is almost quantitative.

Π.

$$Na_2S \cdot 9 H_2O = Na_2S + 9 H_2O$$
240.21 78.06 162.15

Pure  $Na_2S \cdot 9H_2O$  is allowed to stand for 14 days in vacuum over concentrated sulfuric or phosphoric acid. The temperature should be 15°C at the start, later 30 to 35°C. The resulting salt contains only 4% H<sub>2</sub>O. The last traces of water are removed by heating the material to 700°C in a glass or porcelain tube in a fast stream of dry, well-purified H<sub>2</sub>. Granular, pure white Na<sub>2</sub>S of 99.5-99.8% purity is obtained.

PROPERTIES:

White crystals; very hygroscopic; discolors rapidly in moist air. M.p.  $1180 \pm 10^{\circ}$ C (in vacuum); d 1.86.

Readily soluble in water; solutions have a strongly alkaline reaction.

Crystallizes in C1 structure type.

**REFERENCES:** 

- I. W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. 241, 281 (1939).
- II. G. Courtois, Compt. Rend. Séances Acad. Sci. 207, 1220 (1938).

## **Potassium Sulfide**

### K<sub>2</sub>S

# $2 K + S = K_2 S$ 78.19 32.07 110.26

The preparation is carried out in liquid  $NH_3$  in the apparatus shown in Fig. 160.

A weighed tube containing very pure, vacuum-distilled potassium (see section on Alkali Metals) is opened by breaking off the tip and is quickly pushed into side arm A; the latter is immediately capped off. The system is then evacuated through B with a highvacuum pump, and A is heated carefully until the metal melts and flows through the capillary k into the reaction vessel R. Small amounts of oxide which formed on opening the ampoule remain in the latter. After a vigorous stream of dry, carefully purified N<sub>2</sub> is introduced at B, A is reopened and the exact weight of reactant metal is determined by removing and reweighing the ampoule. The stoichiometric quantity of sulfur is then introduced at C, while the passage of N<sub>2</sub> is continued. Vessel R is then cooled with a Dry Ice

360

bath, and 100 ml. of pure  $NH_3$  is condensed on top of both reactants. The reaction forming  $K_2S$  is quantitative if the  $NH_3$  is then allowed to reevaporate over a period of several hours through drying tubes filled with KOH. The ammoniacal solution becomes colorless to-



Fig. 160. Preparation of potassium sulfide.

ward the end. and the K<sub>2</sub>S separates as a white precipitate. As the last traces of NH<sub>3</sub> are removed, the evaporation is slowed down by frequent cooling in order to prevent spattering of the crystals in the reaction vessel. At the end of the run, the system is evacuated for some time with the oil pump and vessel R is heated for 0.5 hour at 400-500°C. The crystals are pulverized by shaking and transferred into storage tubes K by tipping the apparatus. The tubes are then sealed off in vacuum or in a N<sub>2</sub> atmosphere. The K<sub>2</sub>S is very pure; the yield is almost quantitative.

The procedure given above for  $Na_2S$  may also be used for  $K_2S$ .

PROPERTIES:

White crystalline powder; very hygroscopic and sensitive to air. M.p. 912°C; d 1.74.

Readily soluble in water; solutions give a strongly alkaline reaction.

Crystallizes in C 1 structure type.

**REFERENCES**;

- W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. 241, 281 (1939).
- See also J. Goubeau, H. Kolb and H. G. Krall, Z. anorg. allg. Chem. 236, 45 (1938).

# Sodium Disulfide

 $Na_2S_2$ 

I.

$$Na_2S_4 + 2 Na = 2 Na_2S_2$$
  
174.25 45.99 220.24

A solution of  $Na_2S_4$  in absolute alcohol is prepared according to the directions given on p. 365. While the solution is kept warm

and a continuous stream of  $H_2$  or  $N_2$  is passed through, 4 g. of shiny Na, cut into fairly large pieces, is introduced rapidly, one after another, by briefly removing the reflux condenser. The solution is then heated for about 30 minutes at 80°C. The light yellow  $Na_2S_2$  precipitate is rapidly suction-filtered through a glass frit in a vigorous  $H_2$  or  $N_2$  stream. It is thoroughly washed several times with absolute alcohol in order to remove occluded  $Na_2S_4$ and adhering mother liquor. It is then allowed to stand in a vacuum desiccator over  $P_2O_5$ . Even after a fairly long time in the desiccator, the product still contains about 4% of tenaciously adhering  $C_2H_5OH$ . The yield is 7-8 g. of  $Na_2S_2$ .

II.  $Na_2S + S = Na_2S_2$ 78.06 32.07 110.13

About 2.5 g. of Na<sub>2</sub>S and the stoichiometric quantity of S are placed in a Pyrex tube in a vigorous stream of dry; O<sub>2</sub>-free N<sub>2</sub>. After the tube has been sealed off in high vacuum, the reactants are fused at 500°C in an electric furnace until completely homogenized (about 45 minutes are required). The melt solidifies on cooling to form yellow, very hard Na<sub>2</sub>S<sub>2</sub>.

 $2 \text{ Na} + 2 \text{ S} = \text{Na}_2 \text{S}_2$ 45.99 64.13 110.12

This method of preparation from the elements (in liquid  $NH_3$ ) given by Fehér and Berthold, follows in principle the method worked out by Klemm and co-workers for preparing  $Na_2S$  (method I, p. 358).

SYNONYM:

III.

Disodium disulfane.

PROPERTIES:

Light yellow, microcrystalline, very hygroscopic powder. Gradually darkens on heating and at  $400^{\circ}$ C is a light reddish-brown. Above  $475^{\circ}$ C, the color deepens to dark brown and the substance sinters. M.p. ~490°C. The deep brown melt is very mobile. After solidification and thorough grinding in a mortar, only the preparation obtained from liquid ammonia retains the original yellow color. The products prepared from alcohol are always olive green after melting. X-ray studies have shown that sodium disulfide occurs in two modifications, a low-temperature modification

362

 $(\alpha-Na_2S_2)$  and a high-temperature modification  $(\beta-Na_2S_2)$ . The lines of pure  $\alpha-Na_2S_2$  were observed only in preparations isolated from alcoholic solution. Products which solidified from a melt or were exposed to higher temperatures have the lattice of the  $\beta$ -modification. Irreversible transition from one modification to the other occurs between 150 and 250 °C.

**REFERENCES:** 

- I. A. Rule and J. S. Thomas, J. Chem. Soc. (London) <u>105</u>, 177 (1914); T. G. Pearson and P. L. Robinson, J. Chem. Soc. (London) <u>1930</u>, 1473; <u>1931</u>, 1304; F. Fehér and H. J. Berthold, Z. anorg. allg. Chem. <u>273</u>, 144 (1953).
- II. W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. 241, 281 (1939); 225, 273 (1935).
- III. See Na<sub>2</sub>S (I); F. Fehér and H. J. Berthold, Z. anorg. allg. Chem. 273, 144 (1953).

### Potassium Disulfide

### $K_2S_2$

$$K_2S + S = K_2S_2$$
  
110.26 32.07 142.33

The procedure is the same as in method II for  $Na_2S_2$ . On cooling the melt, a hard yellow cake of  $K_2S_2$  is formed.

п.

I.

$$2 K + 2 S = K_2 S_2$$
  
78,19 64,13 142,32

Prepared in liquid NH<sub>3</sub> analogously to method III for Na<sub>2</sub>S<sub>2</sub>.

SYNONYM:

Dipotassium disulfane.

PROPERTIES:

From liquid NH<sub>3</sub>, it is a fine, pale yellow powder which, after heating at 150°C in vacuum for several hours, is free of all NH<sub>3</sub>. Extremely hygroscopic and sensitive to air. The aqueous solution is light yellow. On heating in vacuum,  $K_2S_2$  gradually darkens and at 440°C is deep orange. At higher temperatures, the substance sinters; at 500°C it is a dark red-violet and it melts at about 520°C to form a dark brown fluid. After pulverizing, the solidified melt reverts to the original yellow color. d (20°C) 1.973. Occurs in only one modification.

REFERENCES:

- I. The same as for  $Na_2S_2$  (II). II. The same as for  $Na_2S_2$  (III); F. Fehér and H. J. Berthold, Z. anorg, allg, Chem, 274, 223 (1953).

## Potassium Trisulfide

K<sub>2</sub>S<sub>3</sub>

I.

 $K_2S + 2S = K_2S_3$ 110.26 64.13 174.39

The reaction is carried out in absolute alcohol.

A solution of 5.0 g. of shiny K in 72 ml. of carefully dehydrated ethanol is prepared in a round-bottom flask equipped with a reflux condenser and a drying tube. After the evolution of H<sub>2</sub> ends, exactly half the solution is decanted and is saturated with H<sub>2</sub>S (in the absence of atmospheric moisture) in the manner described for NaHS. Excess H<sub>2</sub>S is removed by brief boiling in a stream of N<sub>2</sub> or H<sub>2</sub>. The two portions of the solution are then recombined in the original reaction flask. This K<sub>2</sub>S solution is then mixed with 4.1 g. of pure S and boiled for 30 minutes; orange-yellow K<sub>2</sub>S<sub>3</sub> crystals separate. These are quickly suction-filtered in a fast  $H_2$  or  $N_2$  stream, washed with absolute ethanol, and freed of adhering solvent in a vacuum desiccator over P2O5.

II. The solid starting materials are melted together. The procedure is the same as in method II for Na<sub>2</sub>S<sub>2</sub>. A brown hard cake of K<sub>2</sub>S<sub>3</sub> is formed on cooling the melt.

SYNONYM:

Dipotassium trisulfane.

PROPERTIES:

Well crystallizing, yellow-orange compound, which is readily soluble in water, giving a yellow solution. On heating, the sub-stance becomes increasingly darker, begins to sinter above 284°C, and melts at about 292°C to form a brown-black melt. After solidification and pulverization, K<sub>2</sub>S<sub>3</sub> has a greenish-brown color. The preparations made by the melt process are of the same color. d (20°C) 2.102.

364

**REFERENCES:** 

- I. T. G. Pearson and P. L. Robinson, J. Chem. Soc. (London) 1931, 1304; the same as for  $K_2S_2$  (II). The same as for Na<sub>2</sub>S<sub>2</sub> (II). II.

Products of the analytical composition Na<sub>2</sub>S<sub>2</sub> may be prepared from the elements either in liquid ammonia or in a melt. However, these products are simply equimolar mixtures of Na<sub>2</sub>S<sub>2</sub> and Na<sub>2</sub>S<sub>4</sub>,

**REFERENCE:** 

F. Fehér and H. J. Berthold, Z. anorg. allg. Chem. 273, 144 (1953).

### Sodium Tetrasulfide

### Na<sub>5</sub>S<sub>4</sub>

I.

2 NaHS	+ 3S =	$Na_2S_4$	+	$H_2S$
112.14	96.20	174.25		34.09

A mixture of 50 ml, of absolute alcohol and about 2 g, of fairly small pieces of shiny Na is prepared in a 150-ml, flask equipped with a reflux condenser and an (initially closed) side arm for holding a gas inlet tube. A drying tube on the condenser prevents contact with atmospheric moisture. After all the Na has dissolved to the ethoxide, a glass tube extending to the bottom of the flask is introduced through the side arm. It is sealed tightly in place, and pure, carefully dried H<sub>2</sub>S is passed through it to saturate the solution. The slight precipitate of NaHS which sometimes forms is not troublesome since it redissolves during the reaction. Then the stoichiometric quantity of pure, very finely powdered S (2.00 g. of Na corresponds to 4.17 g. of S) is added and the reaction mixture boiled for one hour on a steam bath while a vigorous stream of O<sub>2</sub>-free, dry H<sub>2</sub> or N<sub>2</sub> is passed through. A dark-red solution of Na<sub>2</sub>S<sub>4</sub> is formed. This is condensed to 5 ml. in vacuum at about 40°C, causing a dense, yellow precipitate of Na<sub>2</sub>S<sub>4</sub> to separate. The product is rinsed out onto a fritted filter with some absolute alcohol, quickly suction-filtered in a vigorous H<sub>2</sub> or N<sub>2</sub> stream, and washed with some alcohol. On prolonged standing in a vacuum desiccator over  $P_2O_5$ , it loses the adhering alcohol except for small, stubbornly retained quantities (about 2-3%). The yield is 5-6 g.

II.

 $Na_2S + 3S = Na_2S_4$ 78.06 96.20 174.26

The procedure is the same as that described in method II for  $Na_2S_2$ . A grayish-yellow, hard cake of  $Na_2S_4$  is formed on cooling the melt.

SYNONYM:

Disodium tetrasulfane.

PROPERTIES:

Sodium tetrasulfide formed from alcoholic solution is a hygroscopic, orange-yellow crystalline powder. The aqueous solution is yellow at room temperature and dark red at the boiling point. Becomes brown on heating, begins to sinter at 284°C, melts at 286°C to a brown-black mass. After solidification, the substance is green.  $d(20^{\circ}C)$  2.08.

REFERENCES:

The same as for Na<sub>2</sub>S<sub>2</sub> (I and II).

### Potassium Tetrasulfide

K<sub>3</sub>S<sub>4</sub>

I.

$$K_2S + 3S = K_2S_4$$
  
110.26 96.20 206.46

The procedure is the same as in method II for Na<sub>2</sub>S<sub>2</sub>. A hard, wine-red cake of  $K_2S_4$  is formed on cooling the melt.

II.

Preparation in liquid NH3 is analogous to that described in method III for Na<sub>2</sub>S<sub>2</sub>.

SYNONYM:

Dipotassium tetrasulfane.

**PROPERTIES:** 

Orange-yellow solid. The aqueous solution is yellow at room temperature, red at the boiling point. Begins to sinter at 145°C, and melts around 159°C without a sharp melting point. Dark red melt.

366

On cooling, solidifies to a dark-red, glassy mass. In the preparation of the very pure compound from liquid  $NH_3$ , supercooling of the solution can be eliminated by heating it to  $110^{\circ}C$ . The color of the substance is then again orange-yellow but slightly more intense than before the melting.

**REFERENCES:** 

The same as for  $Na_2S_2$  (II) and  $K_2S_2$  (II).

### Sodium Pentasulfide

Na<sub>2</sub>S<sub>5</sub>

 $Na_2S + 4S = Na_2S_5$ 78.06 128.26 206.32

The procedure is the same as in method II for  $Na_2S_2$ . A hard, grayish-yellow cake of  $Na_2S_5$  is formed on cooling the melt.

SYNONYM:

Disodium pentasulfane.

PROPERTIES:

Yellowish-brown powder, very hygroscopic. The microscope reveals uniform, light yellow, partly transparent crystal fragments. If dissolved in freshly boiled water or dilute sodium hydroxide, slight precipitation of S takes place. This disappears almost completely when the solution is heated.

M.p. 253°C; d (20°C) 2.08.

**REFERENCES:** 

The same as for Na<sub>2</sub>S<sub>2</sub> (I and II).

### **Potassium Pentasulfide**

### K<sub>2</sub>S<sub>5</sub>

I.

2 KHS	+ 4S =	$K_2S_5$	+	H <sub>2</sub> S
144.34	128.26	238.52		34.08

In principle, the reaction is the same as that described in method I for  $Na_2S_4$ . The experimental procedure is therefore completely analogous.

Dissolving about 2 g. of K in 30 ml. of anhydrous  $C_2H_5OH$  and subsequent saturation with  $H_2S$  yields an (absolute) alcoholic KHS solution. This solution is then mixed with the stoichiometric quantity of very finely powdered S (2.00 g. of K corresponds to 3.28 g. of S) and is then boiled for one hour on a steam bath while  $H_2$  or  $N_2$  is passed through. The reaction is instantaneous and is accompanied by  $H_2S$  evolution and discoloration of the solution; after a short time, a light orange-red precipitate of  $K_2S_5$  is formed. The quantity of this precipitate increases considerably if the solution is then concentrated in vacuum to about 5 ml. The  $K_2S_5$  is then quickly suction-filtered in a vigorous  $H_2$  or  $N_2$  stream, washed with some absolute ethanol, and freed of adhering solvent in a vacuum desiccator over  $P_2O_5$ . The yield is about 5 g.

II.  $K_2S + 4S = K_2S_5$ 110.26 128.26 238.52

The procedure is the same as described in method II for  $Na_2S_2$ . A hard, dark-brown cake of  $K_2S_5$  is formed on cooling the melt.

SYNONYM:

Dipotassium pentasulfane.

PROPERTIES:

Depending on the method of preparation, orange-red, shiny crystals or dark brown, microcrystalline mass. After a short exposure to air the crystals become coated with a sulfur layer. A solution in  $CO_2$ -free water is yellow to orange-red and becomes deep, dark red at the boiling point. On heating in vacuum, the substance at first becomes dark red and then, at 190°C, deep violet. M.p. 211°C; d (20°C) 2.128.

**REFERENCES:** 

The same as for  $Na_2S_2$  (I and II) and  $K_2S_2$  (II).

### Potassium Hexasulfide

K<sub>2</sub>S<sub>6</sub>

About 5 g. of  $K_2S_5$  is heated together with the required quantity of S for several hours in an evacuated glass tube. The temperature is between 220°C and 280°C, and heating is continued until a completely homogeneous melt is obtained. This procedure is analogous to that described in method II for  $Na_2S_2$ . The melt is allowed to cool slowly over a period of about 10 hours.

SYNONYM:

Dipotassium hexasulfane.

**PROPERTIES:** 

After pulverizing,  $K_2S_6$  has a red to reddish-brown color and no longer forms a clear solution in water. Sintering starts at about  $184^{\circ}C$ . M.p.  $196^{\circ}C$ ; d ( $20^{\circ}C$ ) 2.02.

**REFERENCES:** 

The same as for  $K_2S_2$  (II).

The other alkali metal sulfides  $Rb_2S_2$ ,  $Rb_2S_3$ ,  $Rb_2S_5$ ,  $Cs_2S_2$ ,  $Cs_2S_3$  and  $Cs_2S_6$  can be prepared from the elements in liquid  $NH_3$  in the same way as described in method III for  $Na_2S_2$ .

**REFERENCE:** 

F. Fehér and K. Naused, Z. anorg. allg. Chem. 283, 79 (1956).

#### Ammonium Pentasulfide

### $(NH_4)_2S_5$

 $2 \text{ NH}_{3} + \text{H}_{2}\text{S} + 4 \text{ S} = (\text{NH}_{4})_{2}\text{S}_{5}$ 34.07 34.08 128.26 196.41

A mixture of 80 g. of finely crystalline S in 200 ml. of concentrated ammonium hydroxide (d 0.88, about 35%) is prepared in a one-liter round-bottom flask with a side gas inlet tube reaching to the bottom of the flask. After the closed flask has been weighed, a moderately fast stream of pure, dry  $H_2S$  is passed through the suspension. The neck of the flask carries a one-hole rubber stopper closed off with a plug of absorbent cotton. This flow impediment causes a constant positive pressure within the vessel and thus prevents the inflow of atmospheric  $O_2$ . The solution is at first orange and later becomes dark red; the S dissolves, with occasional shaking, after 60 to 80 minutes. More  $H_2S$  is then introduced until a total of about 35 g. is absorbed. The impurities are then removed by rapid suction-filtering of the dark solution. The filtrate is allowed to stand in a tightly sealed flask at room temperature or on ice until crystallization occurs. After several hours, yellow needles of  $(NH_4)_2S_5$  are formed. These are stable for a fairly long time if submerged in the mother liquor in the absence of air, but they decompose very rapidly when dry to form NH<sub>3</sub>, H<sub>2</sub>S and S. The product is isolated by rapid suctionfiltration through filter paper, removal of adhering mother liquor by pressing between filter papers, and consecutive washing with ether-methanol (5 : 1), absolute ether and anhydrous chloroform. The still moist product is allowed to stand in a vacuum desiccator over CaO which has been wetted with some concentrated ammonium hydroxide. However, because of decomposition, it contains about 10% of elemental sulfur after five hours. Freshly prepared  $(NH_4)_2S_5$  should give a clear solution in 5% ammonium hydroxide.

SYNONYM:

Diammonium pentasulfane.

PROPERTIES:

Yellow to orange-yellow crystals; very easily decomposed to  $NH_3$ ,  $H_2S$  and S. Melts in a sealed tube at 95°C to form a red liquid; decomposes on heating in an open tube. Rapidly precipitates S with water; soluble in ammonium hydroxide (see above).

**REFERENCE:** 

H. Mills and P. L. Robinson, J. Chem. Soc. (London) 1928, 2326.

# Dichloromonosulfane

SCl<sub>2</sub>

I.

$$S + Cl_2 = SCl_2$$
  
32.07 70.91 102.98

Coarsely ground roll sulfur (200 g.) is placed in a one-liter, round-bottom, ground glass flask equipped with a side arm serving as gas inlet. A reflux condenser is set in the ground joint and a thermometer is fastened in such a way that it protrudes from the flask into the lower part of the condenser. From the upper end of the condenser an outlet tube leads to the hood through a wash bottle containing  $H_2SO_4$ . A fast stream of carefully dried  $Cl_2$  gas is passed through the S until the contents of the flask have completely liquefied, forming crude  $S_2Cl_2$  (heat is evolved). Then a

spatula tip (about 0.1 g.) of Fe powder or anhydrous FeCl<sub>2</sub> or FeCl<sub>3</sub> is added and the gas flow is continued for another 0.5 hour; during this time the reaction mixture is gradually cooled to  $20^{\circ}$ C by immersing the flask in water. The dark red liquid which forms, and which contains S<sub>2</sub>Cl<sub>2</sub> and Cl<sub>2</sub> in addition to the SCl<sub>2</sub>, is left to stand for about one hour. Then 2 ml. of PCl<sub>3</sub> is added and the solution is distilled through a small fractionating column. The middle fraction boiling between 55 and 62°C is collected in a receiver containing a few drops of PCl<sub>3</sub> and is again fractionated. A very pure product with a constant boiling point of  $60^{\circ}$ C is obtained. The yield is about 70%.

The substance is stable for a few days when stored in glass vessels in the presence of a few drops of  $PCl_3$ . Pure  $SCl_2$  can always be recovered from the mixtures with  $S_2Cl_2$  and  $Cl_2$  that form on prolonged standing by distillation with  $PCl_3$ .

II.  $\begin{array}{ccc} S_2 Cl_2 \ + \ Cl_2 \ = \ 2 \ SCl_2 \\ 135.04 \ \ 70.91 \ \ 205.95 \end{array}$ 

The substance can also be prepared starting directly with  $S_2Cl_2$ . Otherwise, the procedure is the same as in method I.

SYNONYM:

Sulfur dichloride.

PROPERTIES:

Dark-red liquid with a pungent, chlorinelike odor; it decomposes readily, reversing reaction II, to form  $S_2Cl_2$  and  $Cl_2$ ; sensitive to atmospheric moisture. M.p. --121°C, b.p. +59.6°C; d (20°C) 1.621.

Reacts with water with precipitation of sulfur and formation of  $H_2S_2O_3$ ,  $H_2S_nO_6$  and  $H_2SO_4$ ; soluble in n-hexane without decomposition.

REFERENCES:

 H. Jonas and H. Stöhr, unpublished, private communication.
 See also Naturforschung und Medizin in Deutschland 1939-1946 (FIAT-Review), 23, 191.

# Dichlorodisulfane

 $S_2Cl_2$ 

 $2S + Cl_2 = S_2Cl_2$ 64.13 70.91 135.04

Sulfur is melted in a flask equipped with a side arm and a neck elongated into a gradually narrowing tube. By tilting the flask, the
walls are coated with a uniform layer of sulfur melt. After cooling, the flask is mounted vertically in such a way that the tube end passes through a rubber stopper into another flask below. From the latter, a gas outlet tube passes through a drying tube directly to the hood. A moderately rapid stream of carefully dried  $Cl_2$  is introduced through the side arm of the top flask and the walls of that flask are simultaneously heated to 50-80°C by fanning with a flame. Once the reaction is in progress, the S<sub>2</sub>Cl<sub>2</sub> product flows into the lower flask in a rapid succession of drops. The orangered substance is still contaminated by dissolved starting materials. Some sulfur is added and it is distilled at atmospheric pressure. The portion which distills above 137°C is refractionated over sulfur at about 12 mm. in an apparatus with ground glass joints; b.p. of the pure product is  $29-30^{\circ}C$  at this pressure.

This material is used on a large scale in industry as a solvent for sulfur in the vulcanization of rubber.

### SYNONYMS:

Disulfur dichloride; older designations "sulfurous chloride" and "sulfur monochloride."

### PROPERTIES:

Golden yellow, oily liquid; when less pure, orange to reddish because of SCl<sub>2</sub> impurities; fumes in moist air, unpleasant pungent odor. M.p. -77°C, b.p. (760 mm.) +138°C; d (20°C) 1.6773. Hydrolyzes with water to form HCl, SO<sub>2</sub> and H<sub>2</sub>S; these then

convert to S, H<sub>2</sub>S<sub>2</sub>O<sub>3</sub> and H<sub>2</sub>S<sub>n</sub>O<sub>6</sub>. Readily soluble in CS<sub>2</sub>.

REFERENCE:

Abegg, Handbuch der anorganischen Chemie, Vol. IV, 1, p. 287.

# Dichlorotri-, -tetra-, -tenta-, -hexa-, -hepta- and -octasulfane

 $S_{3}Cl_{2}, S_{4}Cl_{2}, S_{5}Cl_{2}, S_{6}Cl_{2}, S_{7}Cl_{2}, S_{8}Cl_{2}$ 

If the ratios of the reactants are reversed, then the general synthetic method used in the preparation of the sulfanes (p. 353) can also be used for the preparation of the chlorosulfanes. The sulfane component is added to an excess of chlorosulfane at low temperature and after the reaction

$$\begin{array}{l} (a-2) \ Cl_2S_n + Cl - S_n - \boxed{Cl + H} - S_m - \boxed{H + Cl} - S_n - Cl \\ = \ Cl_2S_{2n+m} + 2 \ HCl + (a-2) \ Cl_2S_n \\ (n = 1, \ 2; \ m = 1, \ 2, \ 3, \ 4, \ 5, \ 6). \end{array}$$

is complete, excess chlorosulfane  $(a - 2)Cl_2S_n$  is distilled off. With suitable choice of reactants and careful following of

analogous preparative conditions, it is possible to obtain pure  $S_3Cl_2$ ,  $S_4Cl_2$ ,  $S_5Cl_2$ ,  $S_6Cl_2$ ,  $S_7Cl_2$  and  $S_8Cl_2$ .

In general, the chlorosulfanes are best made from the following components:

Desired chlorosulfane	Chlorosulfane component	Sulfane component	
S <sub>3</sub> Cl <sub>2</sub>	SC12	H <sub>2</sub> S	
S <sub>4</sub> Cl <sub>2</sub>	SCI2	H <sub>2</sub> S <sub>2</sub>	
S <sub>5</sub> Cl <sub>2</sub>	S <sub>2</sub> Cl <sub>2</sub>	H <sub>2</sub> S	
5 5	or SCl <sub>2</sub>	H <sub>2</sub> S <sub>3</sub>	
S <sub>6</sub> Cl <sub>2</sub>	S <sub>2</sub> Cl <sub>2</sub>	H <sub>2</sub> S <sub>2</sub>	
	or SCl <sub>2</sub>	$H_2S_4$	
$S_7Cl_2$	S <sub>2</sub> Cl <sub>2</sub>	H <sub>2</sub> S <sub>3</sub>	
S <sub>8</sub> Cl <sub>2</sub>	S <sub>2</sub> Cl <sub>2</sub>	H <sub>2</sub> S <sub>4</sub>	

The same rules as those for the preparation of sulfanes (p. 349) apply for these procedures, particularly as far as cleaning the glass equipment is concerned.

The syntheses of  $S_3Cl_2$  and  $S_4Cl_2$  are described as examples.

#### Dichlorotrisulfane

S<sub>3</sub>Cl<sub>2</sub>

 $\begin{array}{c} \text{Cl} - \text{S} - \boxed{\text{Cl} + \text{H}} - \text{S} - \boxed{\text{H} + \text{Cl}} \text{S} - \text{Cl} = \text{S}_{3}\text{Cl}_{2} + 2 \text{ HCl} \\ 102.98 & 34.08 & 102.98 & 167.11 & 72.93 \end{array}$ 

#### Excess SC12

The two-liter, three-neck flask a of Fig. 161, equipped with a Hg-seal stirrer, is charged with 2400 g. of freshly distilled SCl<sub>2</sub> and is cooled with stirring to  $-80^{\circ}$ C in a large Dewar flask filled with Dry Ice-methanol mixture; the flask must be completely immersed in the cooling bath. The CaCl<sub>2</sub> tube b prevents penetration of atmospheric moisture. About 45 ml. of previously condensed H<sub>2</sub>S (compare preparation of tetrasulfane, p. 353) is combined with the SCl<sub>2</sub> at about  $-80^{\circ}$ C by attaching the trap filled with liquid H<sub>2</sub>S directly to the somewhat inclined ground glass joint c and emptying it by slow rotation of the male ground glass joint. The reaction, barely visible from the outside, is complete in 12-14 hours; the



Fig. 161. Preparation of dichlorotrisulfane.



Fig. 162. Distillation of dichlorotrisulfane.

cooling bath is then removed and the solution allowed to warm slowly to about  $0^{\circ}$ C with constant stirring.

After most of the dissolved HCl has escaped, an accurately fitting riser tube is placed on the middle ground glass joint d, as shown in Fig. 161. This tube is connected to the side arm of distillation flask e by means of a short piece of rubber tubing. Flask e is then evacuated with an aspirator. The input of the reaction mixture to e is regulated by means of stopcock f. The distillation flask is heated with a small burner and dry, purified N<sub>2</sub> is introduced through the capillary g to aid the boiling. The 500-ml. trap h, cooled with a Dry Ice-methanol bath, can be emptied from time to time into flask i through the sealed-in siphon by briefly releasing the aspirator vacuum. After the bulk of the solvent is removed two hours are needed to distill off two liters of SCl<sub>2</sub> — the product is transferred into the two-neck flask of the high-vacuum apparatus shown in Fig. 162. Here the remaining solvent is removed and collected in the liquid-nitrogen-cooled trap at the pump (not shown in the diagram). Only then is  $\mathcal{J}_1$  cooled with an ice-water bath and  $\mathcal{J}_2$  with a Dry Ice-methanol bath. The flask is then heated on a water bath of 40-50°C; the S<sub>3</sub>Cl<sub>2</sub> slowly distills and mostly condenses in  $\mathcal{J}_1$  (b.p. 30.5°C at 10<sup>-4</sup> mm.).

In this manner, about 100 g. of  $S_3Cl_2$  may be distilled in five hours. The yield of pure  $S_3Cl_2$  is 50%.

A simplified procedure for the continuous preparation of larger quantities of dichlorotrisulfane was developed by S. Fehér and J. Goebell.

### Dichlorotetrasulfane

 $S_4Cl_2$ 

 $\begin{array}{c} \text{Cl} - S - \hline \textbf{Cl} + \textbf{H} \\ 102.98 & 66.15 \\ \hline \textbf{102.98} & 66.15 \\ \hline \textbf{102.98} & 199.18 \\ \hline \textbf{102.98} & \text{Cl} = S_4 \text{Cl}_2 + 2 \text{ HCl} \\ 199.18 & 72.93 \\ \hline \textbf{Excess SCl}_2 \end{array}$ 

The apparatus is the same as that for the preparation of dichlorotrisulfane; only the manner of adding the sulfane component is different.

The three-neck flask a, cooled to  $-80^{\circ}$ C with Dry Ice-methanol mixture, contains 1200 g. of freshly distilled SCl<sub>2</sub>. A separatory funnel, whose tube is drawn into a capillary, is inserted into the ground joint c. About 26 g. of H<sub>2</sub>S<sub>2</sub> is added slowly in drops from this funnel over a period of one to two hours. Good agitation of the flask contents must be provided. The reaction is complete in about three hours.

The rest of the procedure is analogous to that for the preparation of  $S_3Cl_2$ . The residue, a clear, orange-red liquid, is freed of traces of  $SCl_2$  in a smaller flask by short standing in high vacuum; it is then nearly pure so that, in contrast to  $S_3Cl_2$ , distillation is not necessary. The yield is 90%.

The distillation, which involves experimental difficulties, is unnecessary even in the case of higher chlorosulfanes. However, preparations of dichloropenta- and dichlorohexasulfane require about 12 hours for completion, using 25 g. of  $H_2S_3$  and 40 g. of  $H_2S_4$ , respectively.

Reaction of about 450 g. of  $S_2Cl_2$  with approximately 10 g. of  $H_2S$  (10 g. of  $H_2S_2$ ,  $H_2S_3$  or  $H_2S_4$ ) produces  $S_5Cl_2$  ( $S_5Cl_2$ ,  $S_7Cl_2$ ,  $S_8Cl_2$ ). The course of the reaction and the reaction rates are similar to the case in which  $SCl_2$  is used.

PROPERTIES:

Starting with  $S_3Cl_2$ , the higher chlorosulfanes are orange-red, oily liquids with high refractive indexes. They have the same odor as  $S_2Cl_2$  but the intensity of their odor decreases rapidly with increasing molecular weight.

	Formula weight	d (20°C)
S <sub>3</sub> Cl <sub>2</sub>	167.09	1.7441
S <sub>4</sub> Cl <sub>2</sub>	199.15	1.7774
S <sub>5</sub> Cl <sub>2</sub>	231.21	1.8018
S <sub>e</sub> Cl <sub>2</sub>	263.27	1.8219
$S_{7}Cl_{2}$	295.33	1.84
S <sub>8</sub> Cl <sub>2</sub>	327.39	1.85

In this homologous series, the density of the last two members can only be obtained by extrapolation because of their increased viscosity and decreased purity compared to the first members.

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- F. Fehér and J. Goebell, Z. anorg. allg. Chem., in press.

### Sulfur Tetrachloride

SCl<sub>4</sub>

Leg A of the apparatus shown in Fig. 163 is filled to about one third with pure  $SCl_2$  and then sealed off at C. The stoichiometric quantity of  $Cl_2$  is then condensed in the graduated tube B, cooled with a Dry Ice bath. The tube is then sealed off at D. Leg A is then also cooled to  $-78^{\circ}C$  and the apparatus is tilted so that the  $SCl_2$  flows quickly into the  $Cl_2$  in B. The mixture is vigorously shaken. It solidifies immediately, forming white (occasionally pale yellow)  $SCl_4$ .

The product can be stored only in sealed tubes at low temperatures.

The analogous preparation from  $S_2Cl_2$  and liquid  $Cl_2$  is not reccommended since the formation of the intermediate  $SCl_2$  proceeds very slowly in the absence of a catalyst, and the reaction therefore takes several days.



Fig. 163. Preparation of sulfur tetrachloride.

#### **PROPERTIES:**

Fine white powder; stable only as a solid at low temperatures; decomposes above  $-30^{\circ}$ C into SCl<sub>2</sub> and Cl<sub>2</sub>; sensitive to moisture. Solid SCl<sub>4</sub> sinters at  $-30^{\circ}$ C and melts between -30 and  $-20^{\circ}$ C with simultaneous decomposition.

Hydrolyzed by water to HCl and SO<sub>2</sub>.

**REFERENCES:** 

E. Beckmann, Z. phys. Chem. <u>65</u>, 289 (1909). See also O. Ruff, Ber. dtsch. chem. Ges. 37, 4513 (1904).

### Dibromodisulfane

### S<sub>2</sub>Br<sub>2</sub>

I.

2S +	Br <sub>2</sub> =	$S_2Br_2$
64.13	159.83	223.96

The procedure is based on that described by Ruff and Winterfield. Thus, 94 ml. of  $Br_2$  is slowly added to 119 g. of S (recrystallized from  $CS_2$ ) in a dry 500-ml. pressure flask capable of withstanding 20 atm. of internal pressure. (If proper precautions are observed, a beer bottle can be used.) The mixture is heated for two hours on a steam bath. The dark-red liquid product is distilled at 0.1 mm. The first cut contains primarily unreacted  $Br_2$ ; the main fraction distilling between 46 and  $48^{\circ}C$  is pure  $S_2Br_2$ , while crystalline S remains in the distillation flask.

The yield averages 80%.

378 II.

 $S_2Cl_2 + 2 HBr = S_2Br_2 + 2 HCl$ 135.05 161.85 223.96 72.93

The reaction is carried out in the apparatus shown in Fig. 164. Hydrogen bromide is produced by allowing  $Br_2$  to drip into tetralin (see p. 282 for this procedure); to remove traces of elemental bromine, the HBr is passed through a wash bottle containing tetralin, through a U tube containing clay fragments and moist red P, and finally through two  $CaCl_2$  drying tubes. This prepurified gas is then bubbled at room temperature through about 30 g. of dichlorodisulfate (see p. 371), contained in a 250-ml. two-neck flask, in such a way that the contents are well mixed. The course of the slightly exothermic reaction can be followed through the gradual coloration of the oilto a dark red. The reaction is complete after 1-2 hours, as confirmed by qualitative testing of the product for chlorine. During the reaction, the excess HBr and the HCl produced are discharged to the hood via a CaCl<sub>2</sub> tube.



Fig. 164. Preparation of dibromodisulfane.

SYNONYM:

Disulfur dibromide. Older names "sulfur monobromide" and "sulfurous bromide."

PROPERTIES:

Dark red, oily liquid which does not wet glass. M.p.  $-46^{\circ}$ C; dissociates on heating into the elements and therefore can be distilled without decomposition only in high vacuum. d. (20°C) 2.629. In H<sub>2</sub>O, hydrolyzes to HBr, SO<sub>2</sub> and S. Soluble in CS<sub>2</sub>, CCl<sub>4</sub> and C<sub>5</sub>H<sub>6</sub>.

**REFERENCES:** 

- I. O. Ruff and G. Winterfield, Ber. dtsch. chem. Ges. <u>36</u>, 2437 (1903); F. Fehér, J. Kraemer and G. Rempe, Z. anorg. allg. Chem. <u>279</u>, 18 (1955).
- II. F. Fehér and G. Rempe, Z. anorg. allg. Chem. 281, 161 (1955);
   F. Fehér and S. Ristić, Z. anorg. allg. Chem. 293, 311 (1958).

# Dibromotri-, -tetra-, -tenta-, -hexa-, -hepta- and -octasulfane

S<sub>3</sub>Br<sub>2</sub>, S<sub>4</sub>Br<sub>2</sub>, S<sub>5</sub>Br<sub>2</sub>, S<sub>6</sub>Br<sub>2</sub>, S<sub>7</sub>Br<sub>2</sub>, S<sub>8</sub>Br<sub>2</sub>

S <sub>3</sub> Cl <sub>2</sub>	+	2 HBr	=	S <sub>3</sub> Br <sub>2</sub>	+	2 HCl
S <sub>4</sub> Cl <sub>2</sub>	-+-	2 HBr	=	$S_4Br_2$	+	2 HCl
199.18	1	161.85		288.10	1	72.93
S <sub>5</sub> Cl <sub>2</sub> 231.24	+	2 HBr 161.85	=	S <sub>5</sub> Br <sub>2</sub> 320.16	+	2 HCI 72.93
$S_6Cl_2$	+	2 HBr	=	S <sub>6</sub> Br <sub>2</sub>	+	2 HCl
203.31 S <sub>7</sub> Cl <sub>2</sub>	+	2 HBr	=	S <sub>7</sub> Br <sub>2</sub>	+	2 HCl
295.38		161.85		384.29		72.93
S <sub>8</sub> Cl <sub>2</sub>	+	2 HBr	=	$S_8Br_2$	+	2 HCl
327.44		161.85		416.36		72.93

These bromosulfanes are prepared in a manner analogous to  $S_2Br_2$  (method II), using the same apparatus. About 30 g. of chlorosulfane is used.

PROPERTIES:

The color of the bromosulfanes lightens with increasing chain length:  $S_2Br_2$  is dark red,  $S_8Br_2$  only raspberry red. All bromosulfanes are completely miscible with  $CS_2$ . Only the oils containing less sulfur than  $S_4Br_2$  are soluble in benzene, toluene and carbon tetrachloride.

	d (20°C)		
$S_3Br_2$	2.52		
S <sub>4</sub> Br <sub>2</sub>	2.47		
S <sub>5</sub> Br <sub>2</sub>	2.41		
S <sub>6</sub> Br <sub>2</sub>	2.36		
$S_7Br_2$	2.33		
S <sub>8</sub> Br <sub>2</sub>	2.30		
J .			

**REFERENCES:** 

See  $S_2Br_2$  (II).

# Lower Sulfur Oxides

S<sub>2</sub>O, SO

See the original literature for the preparation of these compounds. **REFERENCES:** 

- P. W. Schenk and H. Platz, Z. anorg. allg. Chem. 215, 113 (1933).
- P. W. Schenk and H. Triebel, Z. anorg. allg. Chem. 229, 305 (1936).
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### **Disulfur Trioxide**

### $S_2O_3$

 $S + SO_3 = S_2O_3$ 32.07 80.07 112.14

The Pyrex apparatus (Fig. 165) consists of distillation flasks A, B and C (300 ml. each), reaction vessel D, and U tube K, filled with a  $P_2O_5$ -glass wool mixture. To start the run, about 1 g. of carefully purified S (see p. 341) is charged into vessel D through ground glass joint J, and 200 ml. of pure, 65% oleum is placed in the flask A. Then A is heated very slowly in an  $H_2SO_4$  bath while B is cooled in an ice-water bath and C with an ice-salt bath. Most of the SO<sub>3</sub> condenses in B and only a small amount passes into C. As soon as all the SO<sub>3</sub> which can be removed from the acid by gentle heating is evaporated, flask A is allowed to cool and is then



Fig. 165. Preparation of disulfur trioxide.

disconnected at joint E; the latter is immediately closed off. About three quarters of the SO<sub>3</sub> in B is then distilled into C in the same manner by cooling C with ice water and D with an ice-salt mixture. Flask B is then disconnected at F. If the product collected in C is not yet sufficiently pure (it should melt at 15-16°C to a colorless liquid), it must be redistilled using flasks A and B (which are meanwhile cleaned and dried). To achieve the reaction, the SO<sub>3</sub> in C is heated until it melts, and 15 ml. of the melt is allowed to deposit on the sulfur in D by rotating the flask in the ground glass joint G. The mixture instantaneously turns a deep blue. A vigorous reaction starts after 30 seconds, evolving white vapors. It is then advisable to disconnect the  $P_2O_{\pi}$  tube at H from time to time. It is important to maintain the flask contents at approximately 15°C at all times. Above that temperature, the product S<sub>2</sub>O<sub>3</sub> is markedly decomposed, while at lower temperatures the excess SO<sub>3</sub> solidifies. If this happens, separation of the SO<sub>3</sub> becomes very difficult and is accompanied by partial decomposition of the S<sub>2</sub>O<sub>3</sub>.

The reaction subsides after about two minutes. The mixture is then thoroughly shaken and allowed to stand for five minutes. The solid bluish-green S<sub>2</sub>O<sub>2</sub> settles to the bottom and the almost colorless  $SO_3$  above it can be poured back into flask C by careful tipping. Adhering traces of SO, are removed by subsequent warming in a vigorous CO<sub>2</sub> stream. The initial temperature for this operation is 50°C, which later is reduced to 40°C. When almost all the SO<sub>3</sub> is removed; the contents are rapidly poured into a second dry vessel of the same type and CO<sub>2</sub> is again introduced; the ground glass cap is lifted from time to time and the crystal mass carefully crushed with a glass rod. The surface of the otherwise blue substance begins to turn brown after quantitative removal of the SO<sub>3</sub>. Carbon dioxide is then passed over the product for a further 20 minutes. The temperature should not exceed +10°C. Following this operation, the product has the theoretical composition. The yield is about 3 g.

Disulfur trioxide must be stored below  $+15^{\circ}$ C in a dry, oxygenfree atmosphere at pressures below 1 mm., but even under these conditions it is stable for only a few hours.

PROPERTIES:

Blue-green, crystalline substance; extremely hygroscopic; decomposes readily, particularly above  $+15^{\circ}$ C, into SO<sub>2</sub>, SO<sub>3</sub> and S. Reacts with water with fizzing to form S, H<sub>2</sub>SO<sub>4</sub> and H<sub>2</sub>S<sub>n</sub>O<sub>6</sub>.

Soluble in oleum, giving a deep blue or brown color, depending on the  $SO_3$  content of the acid; insoluble in pure  $SO_3$ .

**REFERENCES:** 

I. Vogel and J. R. Partington, J. Chem. Soc. (London) <u>127</u>, 1514 (1925).

L. Wöhler and O. Wegwitz, Z. anorg. allg. Chem. 213, 129 (1933).

### **Polysulfur Peroxide**

(SO<sub>3-4</sub>)x

See the original literature for the preparation of this compound.

- U. Wannagat and G. Mennicken, Z. anorg. allg. Chem. 268, 69 (1952).
- U. Wannagat and J. Rademachers, Z. anorg. allg. Chem. 286, 81 (1956).

### **Thionyl Chloride**

### $\mathbf{SOCl}_{\mathtt{2}}$

I.

 $SCl_2 + SO_3 = SOCl_2 + SO_2$ 102.98 80.07 118.98 64.07

Flask A of the apparatus in Fig. 166 is charged with pure 65%oleum, flask B with 100 g. of SCl<sub>2</sub> (see p. 370). Flask A is slowly heated in an  $H_2SO_4$  bath, while B is cooled with an ice-water bath. The stoichiometric quantity of SO, is thus gradually distilled onto the SCl<sub>2</sub>. The reaction proceeds with SO<sub>2</sub> evolution (use a hood!). Partial solidification of the flask contents frequently occurs at the beginning. However, the contents should again be completely liquid when the addition is completed. If necessary, the flask is heated at the end for a short time on a water bath. Flask A is then disconnected, the ground joint at C is stoppered, and the mixture is slowly distilled through a column (use a hood!). The mixture must be protected from contact with atmospheric moisture. The middle fraction is further purified by repeated careful fractionation, with S added to the distillation charge in order to convert all sulfur chlorides present to S<sub>2</sub>Cl<sub>2</sub>. The contents are distilled through an efficient column until a completely colorless product, coming over at 76-77°C, is obtained. The yield is about 80% of theoretical.

II. 
$$PCl_5 + SO_2 = POCl_3 + SOCl_2$$
  
208.27 64.07 153.35 118.98

A well-dried, 250-ml., two-neck, round-bottom flask, equipped with a reflux condenser connected to a CaCl<sub>2</sub> tube, is loaded with 100 g. of  $PCl_5$ . Sulfur dioxide is introduced through a gas inlet tube extending to the bottom of the flask. The SO<sub>2</sub> is carefully



Fig. 166. Preparation of thionyl chloride.

predried by allowing it to bubble through two wash bottles containing concentrated  $H_2SO_4$ . The reaction, which can be accelerated by shaking the flask, is complete after several hours, when all the PCl<sub>s</sub> dissolves.

The products are separated by repeated careful fractionation, using an efficient column. B.p. of  $SOCl_2$ ,  $77^{\circ}C$ ; of  $POCl_3$ ,  $108^{\circ}C$ .

This preparation is generally not completely free of phosphorus compounds. The yield of SOC1<sub>2</sub> is 30 g. (50% of theoretical).

**PROPERTIES:** 

Colorless, highly refractive liquid with an unpleasant,  $SO_2$ -like odor. M.p. -104.5°C, b.p. 77°C;  $d_4^2$ ° 1.64. Significant dissociation to  $S_2Cl_2$ ,  $SO_2$  and  $Cl_2$  occurs just above the boiling point. Hydro-lyzes in water to  $SO_2$  and HCl; soluble in benzene and chloroform.

**REFERENCES:** 

I. A. Michaelis, Liebigs Ann. Chem. 274, 173 (1893).

II. H. Grubitsch, Anorganisch-präparative Chemie, Vienna, 1950, p. 294.

## Sulfuryl Chloride

SO<sub>2</sub>Cl<sub>2</sub>

I.

$$SO_2 + Cl_2 = SO_2Cl_2$$
  
64.07 70.91 134.98

a) The reaction proceeds very smoothly and almost quantitatively when activated charcoal is used as the catalyst.

The reactor is a bulb-type condenser with at least six bulbs. Each bulb is approximately half filled with loose glass wool, on top of which there is a layer of granular activated charcoal which occupies less than half of the remaining space. The condenser is clamped vertically; the lower end passes through a rubber stopper and into a suction flask (500-1000 ml.) with its side arm connected to a CaCl, tube in order to protect the contents against atmospheric moisture. Tank Cl<sub>2</sub> and SO<sub>2</sub> are used, if possible, since this is the simplest way to keep the flow rates constant over long periods. The gases are dried separately by passage through wash bottles filled with concentrated  $H_2SO_4$ . These bottles also serve as bubble counters. The gases are then mixed in a Y piece and fed into a flask in which the mixing is completed. When all the air is displaced, the outlet tube of this flask is tightly joined to the upper end of the reactor by means of a rubber stopper. The reactor is well cooled with externally flowing water. At flow rates of 3-4 bubbles/second in both wash bottles, the first SO<sub>2</sub>Cl<sub>2</sub> drops begin to collect in the suction flask after 20-30 minutes. (The induction period can be shortened by moistening the uppermost sphere with a few milliliters of SO<sub>2</sub>Cl<sub>2</sub>.) From then on, about 150 g. of SO<sub>2</sub>Cl<sub>2</sub> is formed per hour. This amount can be increased by increasing the flow rate. The product obtained is quite pure and needs to be fractionated only once at atmospheric pressure, which decreases the yield only slightly.

b) The reaction also proceeds smoothly when camphor is used as the catalyst.

A 500-ml., two-neck, ground-joint flask is provided with a  $CaCl_2$  tube and a gas inlet tube reaching to the bottom. A tee connection allows simultaneous introduction of SO<sub>2</sub> and Cl<sub>2</sub>.

Coarse camphor (10 g.) is placed in the flask, and well-dried  $SO_2$  is passed through until the camphor is saturated and liquefies (1 part of camphor absorbs about 0.88 parts by weight of  $SO_2$  at 725 mm.). Dried  $Cl_2$  is then introduced; it is absorbed by the liquid, forming a colorless solution. After saturation, the  $SO_2$  is reintroduced, followed by renewed passage of  $Cl_2$ . After the formation of approximately 30 g. of sulfuryl chloride, both gases may be introduced simultaneously.

When enough  $SO_2Cl_2$  has been prepared, it is distilled on a water bath. The product obtained in the first distillation still contains camphor, which may be removed by careful fractionation.

 $\begin{array}{rrrr} 2\,\mathrm{HSO_3Cl} &=& \mathrm{SO_2Cl_2} &+& \mathrm{H_2SO_4} \\ 233.06 & & 134.98 & 98.08 \end{array}$ 

An ordinary combustion tube is half filled with pure chlorosulfonic acid (for preparation, see p. 385), is sealed off in the usual

II.

way and is then heated in a paraffin or oil bath for 20 hours at 200-210°C. The initially water-clear liquid takes on a greenish color on cooling. The tube is opened with the usual precautions, the liquid is rapidly poured into a flask, and the fraction boiling below  $110^{\circ}$ C is distilled off. This fraction is redistilled on a water bath and the fraction boiling below 73°C is collected; it consists mainly of SO<sub>2</sub>Cl<sub>2</sub>. To remove HSO<sub>3</sub>Cl and H<sub>2</sub>SO<sub>4</sub>, the product is poured into a separatory funnel filled with crushed ice and is briefly shaken; the lower, cloudy organic layer is removed. This layer is dried for some time in a desiccator over P<sub>2</sub>O<sub>5</sub> and is finally refractionated on a water bath at atmospheric pressure. The middle fraction, boiling between 69 and 70°C, is pure SO<sub>2</sub>Cl<sub>2</sub>. The yield is about 40%.

PROPERTIES:

Colorless, very mobile liquid; yellows slightly on prolonged standing because of partial dissociation; fumes somewhat in air; extremely pungent odor. M.p.  $-54.1^{\circ}$ C, b.p.  $69.2^{\circ}$ C;  $d_{4}^{2\circ}$  1.667.

Decomposes gradually in water to form  $H_2SO_4$  and HCl; under certain circumstances, decomposes explosively with alkalis. Soluble in benzene and toluene.

**REFERENCES:** 

I. a) J. Meyer, Angew. Chem. <u>44</u>, 41 (1931); see also H. Danneel, Angew. Chem. <u>39</u>, 1553 (1926).

b) H. Schulze, J. prakt. Chem. 24, 168 (1881).

II. P. Behrend, J. prakt. Chem. 15, 23 (1877).

### Chlorosulfonic Acid

HSO<sub>3</sub>Cl

$$SO_3 + HCl = HSO_3Cl$$
  
80.07 36.47 116.54

A round-bottom flask, closed with a three-hole rubber stopper, is half filled with oleum of the highest  $SO_3$  concentration possible. A gas inlet tube, reaching to the bottom of the flask, passes through one of the holes of the stopper, a thermometer through the second and a downward condenser ending in a CaCl<sub>2</sub> tube through the third. At the beginning of the run, the flask is inclined so that the condenser points upward. Thus, all products condensing during the reaction flow back into the flask. A slow, carefully dried HCl stream (see p. 280) is then passed through the oleum at room temperature until no further absorption takes place. The condenser is then returned to its downward position and the flask contents are distilled in the HCl stream. The reaction must be protected against atmospheric moisture. The fraction coming over between 145 and  $160^{\circ}$ C is refractionated in the same manner in a clean apparatus with ground glass connections. A colorless distillate, stable for long periods of time, can be obtained only in the complete absence of organic substances (rubber, cork, dust, etc.). The boiling point of the pure middle fraction is  $151-152^{\circ}$ C. The yield is almost quantitative, based on the SO<sub>8</sub> content of the oleum.

Because of slight dissociation on heating, the product contains some dissolved HCl and SO<sub>3</sub>, as well as traces of SO<sub>2</sub>,  $Cl_2$ ,  $SO_2Cl_2$ and  $H_2SO_4$  formed from the above. Completely pure HSO<sub>3</sub>Cl is obtained by fractional crystallization in liquid nitrogen in complete absence of moisture. (For more details, see the original literature.)

PROPERTIES:

Colorless liquid, fuming strongly in moist air; pungent odor. M.p. -80°C, b.p. 152°C; d (20°C) 1.79.

Decomposes violently with water to form HCl and H<sub>2</sub>SO<sub>4</sub>.

**REFERENCES:** 

C. R. Sanger and E. R. Riegel, Z. anorg. allg. Chem. <u>76</u>, 79 (1912). See also H. Beckurts and R. Otto, Ber. dtsch. chem. <u>Ges. 11</u>, 2058 (1878).

A. Simon and G. Kratsch, Z. anorg. allg. Chem. 242, 369 (1939).

### Pyrosulfuryl Chloride

S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub>

The apparatus, consisting of a one-liter flask equipped with a long reflux condenser topped with a drying tube, is set up under an efficient hood because of the ensuing  $\text{COCl}_2$  evolution. Commercial  $\text{SO}_3$  (300 g.) is placed in the flask and  $\text{CCl}_4$  (572 g.) is poured over it. The reaction rate is very low while the flask contents are cold. The flask is carefully heated on a sand bath until all the  $\text{SO}_3$  dissolves and the brown liquid which forms no longer evolves any  $\text{COCl}_2$ . The reaction usually ends in two hours. The contents of the flask are then slowly distilled through a column protected from atmospheric moisture (use a hood!) and the fraction boiling between

135 and 160°C, consisting of crude S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub> contaminated with some HSO<sub>s</sub>Cl, is collected separately. To purify the compound, small pieces of ice are added while the flask is rapidly rotated and efficiently cooled with an ice-salt bath; the ice vigorously hydrolyzes the HSO<sub>3</sub>Cl, evolving HCl, while the S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub> is not attacked to any significant extent. As soon as the gas evolution subsides, the addition of ice is stopped, the flask is allowed to stand for several hours in the cooling bath, and the cold liquid, which by then has separated into two layers, is poured into a separatory funnel. Here the S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub> may become either the upper or the lower layer since the densities of the two phases are very similar. However, a separation with  $S_2O_5Cl_2$  collecting in the bottom layer is easily attained by addition of some concentrated H<sub>2</sub>SO<sub>4</sub> or cold water. The product is dried with  $P_2O_5$  and distilled, while carefully protected from atmospheric moisture, at atmospheric pressure. The fraction distilling between 150 and 153°C is refractionated at reduced pressure. The yield is about 240 g. of pure S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub>.

PROPERTIES:

Colorless, very mobile liquid; characteristic odor; fumes faintly in moist air and becomes cloudy because of the separation of sulfuric acid. M.p.  $-37^{\circ}$ C, b.p.  $152-153^{\circ}$ C with slight decomposition; dissociates into SO<sub>3</sub>, SO<sub>2</sub> and Cl<sub>2</sub> on prolonged refluxing or heating to  $250^{\circ}$ C; d  $_{4}^{2\circ}$  1.84; d  $_{4}^{\circ}$  1.87. Hydrolyzes in water, particularly when warm, to H<sub>2</sub>SO<sub>4</sub> and

Hydrolyzes in water, particularly when warm, to  $H_2SO_4$  and HC1.

**REFERENCES:** 

- W. Prandtl and P. Borinski, Z. anorg. allg. Chem. 62, 24 (1909).
- C. R. Sanger and E. R. Riegel. Z. anorg. allg. Chem. <u>76</u>, 79 (1912).

# Thionyl Bromide SOBr<sub>2</sub>

I.

SOCl <sub>2</sub>	+	2 HBr	=	SOBr <sub>2</sub>	+	2 HC'
118.98		161.85		207,90		72,93

Pure SOCl<sub>2</sub> (50 ml.) (see p. 382) is placed in a 150-ml. ground glass flask provided with a gas inlet tube reaching to the bottom and sealed in at the side. The flask is equipped with a reflux condenser closed off with a  $CaCl_2$  drying tube. A moderately rapid stream of carefully dried HBr is bubbled through for 12 hours while the flask is cooled with ice. The contents gradually turn reddish. At the end of the reaction, the product mixture is distilled in ground glass apparatus at 62 mm.; about 50 ml. of orange-red crude  $SOBr_2$  distills between 69 and 70°C. This crude is refractionated at 20 mm.; b.p. of the pure orange-yellow substance is 48°C. The yield is nearly quantitative.

Stored in sealed glass ampoules or in flasks with very tight ground glass stoppers.

### Other preparative methods:

II. Reaction of  $SO_2$  with  $Br_2$  in the presence of  $PCl_3$ : 64 g. of  $SO_2$  is added to a mixture of 138 g. of  $PCl_3$  and 160 g. of  $Br_2$ , while the reactor is cooled. The yield, after fractional distillation, is 180 g. of  $SOBr_2$ .

III. When  $\tilde{1}$  mole of SOCl<sub>2</sub> is added dropwise to a solution of 2.1 moles of KBr in 150 ml. of liquid SO<sub>2</sub>, SOBr<sub>2</sub> separates as a white precipitate. After evaporation of the SO<sub>2</sub>, the product is distilled at 20°C and 0.1 mm. into a trap at -80°C. The yield is about 50%.

PROPERTIES:

Orange-yellow liquid. On prolonged standing, slowly decomposes into SO<sub>2</sub>, Br<sub>2</sub> and S<sub>2</sub>Br<sub>2</sub> and turns reddish; very sensitive to moisture. M.p.  $-49.5^{\circ}$ C. Its thermal stability is poor and therefore SOBr<sub>2</sub> can be distilled only in vacuum. d. (20°C) 2.685.

Hydrolyzes in water to HBr and SO<sub>2</sub>. Soluble in  $CS_2$ ,  $C_6H_6$ ,  $CHCl_3$  and  $CCl_4$ .

**REFERENCES:** 

- I. H. Hibbert and J. C. Pullman in: H. S. Booth, Inorg. Syntheses, Vol. I, p. 113, New York-London 1939; H. A. Mayes and J. R. Partington, J. Chem. Soc. (London) <u>1926</u>, 2594; see also F. Govaert and M. Hansens, Natuurwetensch. Tijdschr. <u>20</u>, 77 (1938).
- II. German Pat. 665061 (1936) I. G. Farbenindustrie.
- III. M. J. Frazer and W. Gerrard, Chem. and Ind. 1954, 280.

# Peroxymonosulfuric Acid

# H<sub>2</sub>SO<sub>5</sub>

A slight excess of 100% (or at least nearly anhydrous)  $H_2O_2$  (see p. 140) is slowly added to pure, thoroughly cooled chloro-

388

sulfonic acid (see p. 385). Vigorous evolution of HCl occurs. When all the H<sub>2</sub>O<sub>2</sub> has been added and the gas evolution has subsided, the reaction mixture is gradually warmed and the dissolved or still evolving HCl is removed in aspirator vacuum. The mixture cannot be allowed to stand too long prior to HCl removal because the  $H_2SO_5$  tends to oxidize the HCl to Cl<sub>2</sub> and to oxides of chlorine. The HCl-free liquid can then be crystallized in a well-sealed flask. If, after standing for some time at room temperature, it is still not solid, the flask is placed in a cooling bath. After 12 hours the crystals are separated from the mother liquor either by rapid suction filtration on a glass frit or by centrifuging in the apparatus shown in Fig. 103 (p. 141). The effectiveness of this separation determines the purity of the product, which is normally 94-97%. The impurities present are H<sub>2</sub>S<sub>2</sub>O<sub>8</sub> and traces of Cl. The substance can be further purified by partial remelting. The yield is 50-70%. A small additional fraction can be obtained from the mother liquor. The handling of large quantities of H<sub>2</sub>SO<sub>5</sub> is somewhat dangerous. Local overheating, caused, for example, by addition of H<sub>2</sub>O, can cause explosive decomposition. Safety glasses should always be worn.

SYNONYMS:

Hypersulfuric acid, Caro's acid.

**PROPERTIES:** 

Colorless, beautiful crystals; hygroscopic; pure material stable for a few days, although with slight loss of active oxygen, the rate of decomposition is catalysed by the impurities formed; vigorous oxidizing agent. M.p.  $+45^{\circ}$ C with slight decomposition.

Small amounts of  $H_2SO_5$  dissolve in ice water without evolving oxygen or hydrolyzing. When larger amounts are used and the water is at room temperature,  $H_2O_2$  and  $H_2SO_4$  are formed. Readily soluble in alcohol and ether.

**REFERENCES:** 

J. D'Ans and W. Friederich, Ber. dtsch. chem. Ges. <u>43</u>, 1880 (1910); Z. anorg. allg. Chem. <u>73</u>, 325 (1912).

### Peroxydisulfuric Acid

### $H_2S_2O_8$

I.

 $2 \operatorname{HSO}_{3}\operatorname{Cl} + \operatorname{H}_{2}\operatorname{O}_{2} = \operatorname{H}_{2}\operatorname{S}_{2}\operatorname{O}_{8} + 2 \operatorname{HCl}_{233,06}$   $34.03 \quad 194.15 \quad 72.93$ 

Pure chlorosulfonic acid (see p. 385) is mixed slowly and with efficient cooling with the stoichiometric quantity of 100% H<sub>2</sub>O<sub>2</sub>

(see p. 140 for preparation). The reaction proceeds with HCl evolution, which continues for some time after the end of the addition. When the gas evolution subsides, the solution is gradually heated, and both the dissolved and evolving HCl are removed with an aspirator. The liquid is allowed to stand in a closed flask; it usually solidifies slowly after some time at room temperature. The crystallization can be accelerated by cooling or seeding. The HCl evolution resumes during crystallization because the mother liquor becomes enriched with HSO<sub>3</sub>Cl and H<sub>2</sub>SO<sub>5</sub>. After 12 hours, the solid acid is quickly suction-filtered through a glass frit or is isolated by centrifuging in the apparatus shown in Fig. 103 (p. 141). The purity is 92-98%; the yield is 60% of theoretical. A fair amount of a somewhat less pure fraction may still be obtained from the mother liquor by lower cooling.

II. Aqueous solutions of  $H_2S_2O_8$  are obtained by electrolysis of concentrated  $H_2SO_4$  solutions at a high current density and low temperature. For details, see the original literature.

SYNONYMS:

Hypersulfuric acid, persulfuric acid.

PROPERTIES:

Colorless, finely crystalline substance; ozone odor; extremely hygroscopic; pure  $H_2S_2O_8$  is stable for several weeks with only a slight loss of active oxygen, while the impure acid is considerably less stable; strong oxidizing agent. M.p.  $65^{\circ}C$  (dec.).

Dissolves in water, with fizzing and considerable decomposition, to yield  $H_2SO_5$  and  $H_2O_2$ ; soluble without decomposition in alcohol; less soluble in ether.

**REFERENCES:** 

- I. J. D'Ans and W. Friederich, Ber. dtsch. chem. Ges. <u>43</u>, 1880 (1910); Z. anorg. allg. Chem. <u>73</u>, 325 (1912).
- II. K. Elbs and O. Schönherr, Z. Elektrochem. 2, 245 (1895).

### Ammonium Peroxydisulfate

 $(NH_4)_2S_2O_8$ 

Prepared by anodic oxidation of a saturated  $(NH_4)_2SO_4$  solution:

$$2(NH_4)_2SO_4 - 2 e = (NH_4)_2S_2O_8 + 2NH_4^+$$
  
264.29 228.21

A clay cup with a capacity of 130 to 150 ml. is placed in a 500ml. beaker of equal height. The cup is surrounded with a coil of lead tubing which serves both as a cathode and a cooling coil. The current lead is a copper wire soldered to the coil. A 0.05-cm.thick Pt wire spiral anode, ignited prior to the run, is suspended in the middle of the cup. A mixture of equal weights of concentrated  $H_2SO_4$  and  $H_2O$  is used as the cathode liquid; the anode space is filled with an iron-free  $(NH_4)_2SO_4$  solution, saturated at 10°C [76.3 g. of  $(NH_4)_2SO_4$  in 100 g. of  $H_2O$ ].

The cooling water flow is turned on, and the electrodes are connected through a rheostat and an ammeter to the 12-14 v. power supply. The current should be 2.5 amp. and the electrolyte temperature as low as possible. Slight evolution of  $O_2$  can be observed at the anode; simultaneously, an ozone odor is apparent. After 4-6 hours, white crystalline  $(NH_4)_2S_2O_8$  separates in the clay cup. The electrolysis is continued for some time and then the salt is suction-filtered on fritted glass. The remaining mother liquor is largely removed by pressing on clay. The product still contains a few percent of sulfate. To purify the compound, it is recrystallized (from as small an amount of water as possible) by dissolving at  $40^{\circ}C$  and then cooling an ice bath. The pure substance (negative test with BaCl<sub>2</sub>) is dried on clay and in the desiccator. After 10 hours of electrolysis the yield is 33 g.

The analyte liquid, saturated with  $(NH_4)_2S_2O_8$ , can be remixed with fresh  $(NH_4)_2SO_4$  and reelectrolyzed. The catholyte is gradually neutralized by  $NH_4^+$  and therefore must be tested with pH paper and replaced with fresh acid when necessary. After 3-4 hours, another 20-40 g. of  $(NH_4)_2S_2O_8$  is obtained.

### SYNONYM:

The older name is ammonium persulfate.

### PROPERTIES:

Colorless, platelike or prismatic crystals; stable for months when pure and dry; decomposes in the presence of moisture, gradually evolving ozone-containing oxygen; strong oxidizing agent. Decomposes on heating, evolving  $O_2$  and forming  $(NH_4)_2S_2O_7$ . d 1.98.

Solubility  $(0^{\circ}C)$ : 58.2 g.; (15.5°C) 74.8 g./100 g. H<sub>2</sub>O. The solution decomposes slowly at room temperatures and rapidly at higher temperatures, evolving O<sub>2</sub> and forming NH<sub>4</sub>HSO<sub>4</sub>.

Monoclinic crystals, space group  $C_{2h}^5$ .

**REFERENCES:** 

- E. Müller, Elektrochemisches Praktikum [Electrochemical Practice], 7th Ed., Dresden-Leipzig, 1947, p. 212.
- K. Elbs, J. prakt. Chem. [2] 48, 185 (1893).

# Potassium Peroxydisulfate

## $K_2S_2O_8$

Prepared by electrolytic oxidation of saturated KHSO<sub>4</sub> solution:

 $\begin{array}{rcrcrcr} 2 \text{ KHSO}_4 & - & 2 & \mathcal{C} & = & \text{K}_2 \text{S}_2 \text{O}_8 & + & 2 \text{ H}^+ \\ & & 272.34 & & 270.32 \end{array}$ 

A 500-ml. battery jar, placed in a larger container which has an inlet and an outlet for cooling water, is used as the electrolysis vessel. The  $1.4 \times 4$  cm. anode, a shiny Pt sheet, is suspended in the middle. Two Pt wire-gauze cathodes having over-all surface areas of 15 cm.<sup>2</sup> each are attached parallel to the anode, one on either side. The cathode-anode distance is 1.5 cm.

Before the run, the anode is ignited and the jar filled with dilute  $H_2SO_4$  saturated with  $KHSO_4$ . The electrodes are then connected through a rheostat and an ammeter with an 8-12 v. power supply. The anode current is adjusted to a density of 0.48 amp./cm.<sup>2</sup> (5.3 amp, for the given anode surface) and the electrolysis is carried out for a few hours with intensive external cooling. The tempera-ture of the electrolyte should not exceed  $+7^{\circ}$ . The liquid turns cloudy after 10-15 minutes, due to separation of K<sub>2</sub>S<sub>2</sub>O<sub>8</sub>. The salt gradually collects at the bottom of the vessel as a white, loose, very fine crystalline precipitate. It is suction-filtered (good vacuum) on fritted glass and washed with some water. Small amounts of occluded sulfuric acid are removed by repeated recrystallization from water at 30°C. The pure substance (negative test with BaCl<sub>2</sub>) is dried by pressing on clay and in a desiccator over concentrated H<sub>2</sub>SO<sub>4</sub> or CaCl<sub>2</sub>. After three hours of electrolysis, the yield is 27 g., corresponding to an electrolytic yield of 34%.

SYNONYM:

The older name is potassium persulfate.

**PROPERTIES:** 

Colorless prisms or platelike crystals; only the completely dry material is stable for long periods; decomposes in moist air, forming  $\rm KHSO_4$ ; strong oxidizing agent. Evolves oxygen on heating. d 2.477.

Solubility in water (0°C): 1.62 g.; (10°C) 2.60 g.; (20°C) 4.49 g.; (30°C) 7.19 g. of  $K_2S_2O_3/100$  g. of  $H_2O$ . The solution decomposes on prolonged standing, evolving  $O_2$  and forming KHSO<sub>4</sub>.

Triclinic crystals, space group  $C_2^1$ .

REFERENCE:

E. Müller and O. Friedberger, Z. Elektrochem. 8, 230 (1902).

### Cobalt Sulfoxylate

 $C_0SO_2 \cdot 3 H_2O$ 

 $\begin{array}{ccc} \text{Co}(\text{CH}_{3}\text{COO})_{2} &+ & \text{Na}_{2}\text{S}_{2}\text{O}_{4} &= & \text{Co}\text{S}_{2}\text{O}_{4} &+ & \text{Na}\text{CH}_{3}\text{COO} \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & \\ & & & & & \\ & & & & & \\ & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & \\ & & & & & \\ & & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & & \\ & & & & & \\ & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & & \\ & & & & & \\ & & & & & & \\ & &$ 

A solution of 10 g. of  $Co(CH_3COO)_2 \cdot 4H_2O$  in 60 ml. of water is mixed with 10.2 g. of solid  $Na_2S_2O_4^*$  (Co :  $Na_2S_2O_4 = 1 : 1.5$ ) while a fast stream of oxygen-free  $N_2$  is passed through. A solution of 3.5 g. of  $NaHCO_3$  in 50 ml. of  $H_2O$  (Co :  $NaHCO_3 = 1 : 1.1$ ) is then gradually added in small portions. The cobalt salt solution turns red-brown, and a brown, finely powdered precipitate of  $CoSO_2 \cdot$  $3H_2O$  separates, while  $CO_2$  is evolved. The precipitate is rapidly suction-filtered in a nitrogen atmosphere, washed with water, alcohol and ether, and dried on clay in an evacuated desiccator previously flushed with  $N_2$ .

**PROPERTIES:** 

Brown powder. Decomposed by atmospheric oxygen or heat, forming cobalt sulfide. Soluble in  $NH_3$ , pyridine and ethylenediamine, forming deep dark-red solutions.

**REFERENCE:** 

R. Scholder and G. Denk, Z. anorg. allg. Chem. 222, 17 (1935).

### Sodium Dithionite

### $Na_2S_2O_4 \cdot 2 H_2O$

Sodium dithionite dihydrate is prepared by salting out an aqueous solution of commercial  $Na_2S_2O_4$ .

<sup>\*</sup>This refers to 100%  $Na_2S_2O_4$ ; commercial products are usually less pure.

A 20-25% solution of the purest obtainable  $Na_2S_2O_4$  in air-free distilled  $H_2O$  is prepared in an inert gas atmosphere. The solution is filtered in the absence of air. Then 30 g. of finely powdered NaCl per 100 ml. of  $H_2O$  is quickly added and immediately vigorously shaken. After about half a minute, the dihydrate precipitates as a thick, white crystalline slurry. It is suction-filtered (again in the absence of air) and washed with saturated NaCl solution, then with aqueous and finally with anhydrous acetone. The crystals are dried on clay in an evacuated desiccator preflushed with nitrogen. Other salting-out agents include NaHSO<sub>3</sub>, NaOH, NaNO<sub>2</sub>, CH<sub>3</sub>COONa, MgCl<sub>2</sub>, CaCl<sub>2</sub> and ZnCl<sub>2</sub>.

Because of its instability, the dihydrate has no practical uses; anhydrous  $Na_2S_2O_4$  is used on a large scale as a reducing agent in the dye industry, for preparing rongalite ( $CH_2O \cdot Na \cdot HSO_2 \cdot H_2O$ ) and as an  $O_2$  absorbent.

SYNONYM:

The older designation is "sodium hyposulfite," while in industry it is called (incorrectly) "sodium hydrosulfite."

**PROPERTIES:** 

Formula weight 192.13. Colorless, needle-shaped crystals; very air sensitive; decomposes particularly easily when moist, forming  $Na_2S_2O_5$  and  $Na_2S_2O_3$ ; strong reducing agent. Decomposes thermally to  $Na_2S_2O_3$ ,  $Na_2SO_3$  and  $SO_2$ .

Readily soluble in water (2.18 g. of  $Na_2S_2O_4 \cdot 2H_2O/100$  g. of  $H_2O$  at 20°C); insoluble in ethanol.

**REFERENCE:** 

K. Jellinek, Z. anorg. allg. Chem. 70, 93 (1911).

### Zinc Dithionite

ZnS<sub>2</sub>O<sub>4</sub>

 $Zn + 2 SO_2 = ZnS_2O_4$ 65.38 128.13 193.51

A two-liter round-bottom flask with a side arm reaching to the bottom and serving as gas inlet tube is used as the reactor. The flask carries an Anschütz attachment with a mercury-seal stirrer in one opening, while the other is closed with a one-hole rubber stopper with a cotton plug in the hole. In this manner a steady, positive  $SO_2$  pressure is maintained in the flask.

The flask is charged with 750 ml. of ethanol, 250 ml. of  $H_2O$ and 270 g. of finely powdered, high-grade Zn dust. With vigorous stirring, 470 g. of pure  $SO_2$  is introduced at  $60^{\circ}C$ . The reaction is exothermic. After a short time, a paste of coarsely crystalline  $ZnS_2O_4$  begins to separate. When the reaction is complete, the mixture is allowed to cool and is then rapidly suction-filtered in a fast  $H_2$  stream. The crystals are washed with absolute ethanol and dried in vacuum at  $60-70^{\circ}C$ . Acetone may be used for the washing. The yield is nearly quantitative but depends essentially on the composition of the zinc dust. The latter may be activated, if desired, by pretreatment with very dilute  $AgNO_3$  solution.

SYNONYM:

The older designation is "zinc hyposulfite," while in industry it is frequently called "zinc hydrosulfite."

PROPERTIES:

White, crystalline powder;  $SO_2$  odor, due to decomposition on standing in air ( $SO_2$  evolution); strong reducing agent. Readily soluble in water (the ratio is about 1 : 7) with a strong tendency to form supersaturated solutions.

**REFERENCES:** 

German patent 218192 (1907) Badische Anilin- und Sodafabrik; Chemiker-Ztg. Rep. <u>31</u>, 324 (1907).

### Sodium Dithionate

 $Na_2S_2O_6 \cdot 2 H_2O$ 

Ι.

 $MnO_2 + 2SO_2 = MnS_2O_6$ 86.93 128.13

 $\frac{MnS_2O_6 + Na_2CO_3}{(2 H_2O)} = \frac{Na_2S_2O_6}{(2 H_2O)} + \frac{MnCO_3}{114.94}$ 

A one-liter round-bottom flask equipped with a stirrer, a thermometer and a gas inlet tube reaching to the bottom is used as the reactor. The flask is cooled in an ice bath and 500 ml. of  $H_2O$  is saturated with very pure  $SO_2$ . While the contents are vigorously stirred and the addition of  $SO_2$  is continued, 80 g. of finely powdered, very pure  $MnO_2$  is added in portions of 1-2 g. over a period of 2.5-3 hours; the temperature of the mixture should not exceed  $10^{\circ}$ C. The stirring is continued for some time after the addition until there is no further color change. Excess SO<sub>2</sub> is removed in vacuum, while the flask is gently heated to  $40^{\circ}$ C. The gelatinous residue is filtered and washed with warm water.

The filtrate is combined with the wash water and stirred at  $35-40^{\circ}$ C with solid BaCO<sub>3</sub> until there is no further evolution of CO<sub>2</sub>. The stirring is continued for another 10 minutes. The mixture is then neutralized to litmus with solid Ba(OH)<sub>2</sub>. To test for completeness of removal of sulfate and sulfite, a filtered sample of the liquid is mixed with dilute HCl and BaCl<sub>2</sub> solution. If a precipitate is still formed, more hot saturated Ba(OH)<sub>2</sub> solution is added and the test is repeated. When the result is negative, the solution is suction-filtered and the precipitate washed with 50 ml. of water.

Approximately 65 g. of  $Na_2CO_3$  is slowly added, in portions of 1-2 g., to the filtrate at 35°C. The mixture is vigorously stirred and the temperature increased to 45°C. As soon as a continuous test with litmus paper indicates a lasting, definitely alkaline reaction, addition of the  $Na_2CO_3$  is interrupted and the mixture is suction-filtered and washed with 150 ml. of 50°C water containing some  $Na_2CO_3$ . The filtrate is retested with litmus paper and, if necessary, mixed with further  $Na_2CO_3$  and filtered. The solution is concentrated on the water bath to a much smaller volume (discarding any precipitate which might separate at the beginning) and is then allowed to stand for some time at 10°C. The separated  $Na_2S_2O_6 \cdot 2H_2O$  is suction-filtered (good suction, no washing) and dried by pressing on clay. Concentrating the solution too much causes contamination with  $Na_2CO_3$ . The yield is 88%, based on  $MnO_2$  used.

II.

a)  $BaS_2O_6 + Na_2CO_3 = Na_2S_2O_6 + BaCO_3$ (2 H<sub>2</sub>O) (2 H<sub>2</sub>O) 333.52 106.00 242.16 197.37 b)  $BaS_2O_6 + Na_2SO_4 = Na_2S_2O_6 + BaSO_4$ (2 H<sub>2</sub>O) (2 H<sub>2</sub>O) 333.52 142.05 242.16 233.42

A hot solution of  $BaS_2O_6 \cdot 2H_2O$  is mixed with the stoichiometric quantity of  $Na_2CO_3$  or  $Na_2SO_4$ , also dissolved in hot water. After boiling for several hours, the precipitate is filtered off and the solution concentrated. Isolation and drying of the crystals is the same as in method I.

PROPERTIES:

Colorless, water-clear crystals; very stable in air. On heating, the water of crystallization is given off between 60 and  $100^{\circ}$ C;

above  $200^{\circ}$ C quantitative decomposition into Na<sub>2</sub>SO<sub>4</sub> and SO<sub>2</sub>. d 2.19.

Readily soluble in water (at  $0^{\circ}$ C, 6.05; at  $20^{\circ}$ C, 13.39; at  $30^{\circ}$ C, 17.32 wt. % of salt is dissolved); insoluble in alcohol.

**REFERENCES:** 

- I. R. Pfanstiel in: W. C. Fernelius, Inorg. Syntheses, Vol. II, p. 170, New York-London, 1946.
- II. W. C. de Baat, Rec. Trav. Chim. Pays Bas 45, 237 (1926).

#### **Barium Dithionate**

 $BaS_2O_6 \cdot 2 H_2O$ 

 $MnO_2 + 2SO_2 = MnS_2O_6$ 86.93 128.13

A solution of  $MnS_2O_6$  is prepared from sulfurous acid and  $MnO_2$ in the manner described for  $Na_2S_2O_6 \cdot 2H_2O$  (see above). The filtered clear solution is heated to  $25-40^{\circ}C$  and stirred with 160 g. of Ba(OH), • 8H<sub>2</sub>O over a period of 30 minutes. Stirring is continued for 30 minutes after completion of the reaction. The solution is then heated to  $65-75^{\circ}$ C and enough base is added to make the mixture strongly alkaline. Very vigorous agitation is necessary during the addition of the base. The heating and agitation are continued for another 30 minutes. The separated hydrated oxide is filtered hot and washed with 300 ml. of Ba(OH)<sub>2</sub> solution held at 65°C. The wash water is combined with the filtrate and retested for alkalinity. If the alkaline reaction is weak, more Ba(OH)<sub>2</sub> • 8H<sub>2</sub>O is added and the solution is refiltered. The excess base is then precipitated with CO<sub>2</sub> and filtered off. The solution is concentrated on a steam bath to about 50 ml. and cooled to crystallize. The  $BaS_2O_8 \cdot 2H_2O$ precipitate is suction-filtered (the best vacuum possible-no washing!) and freed of traces of mother liquor by pressing on clay. Additional salt may be precipitated from the mother liquor by the addition of 75 ml. of ethanol. The yield is 73%, based on MnO, used.

#### PROPERTIES:

Colorless, monoclinic, prismatic crystals; stable in air. On heating, the water of crystallization is given off at 120°C. Significant

decomposition starts at  $140^{\circ}$ C (SO<sub>2</sub> evolution and formation of BaSO<sub>4</sub>).

Readily soluble in water (at  $0^{\circ}$ C, 7.86; at  $20^{\circ}$ C, 15.75; at  $30^{\circ}$ C, 19.86 wt. % of salt is dissolved); insoluble in alcohol.

**REFERENCE:** 

R. Pfanstiel in: W. C. Fernelius, Inorg. Syntheses, Vol. II, p. 170, New York-London, 1946.

### **Potassium Trithionate**

#### $K_2S_3O_6$

I.

2 KHSO <sub>3</sub>	+ SCl <sub>2</sub> =	K <sub>2</sub> S <sub>3</sub> O <sub>6</sub>	+	2 HCl
240.34	102,98	270.39		72.93

Sulfur dioxide is bubbled through 800 ml. of cooled 5M potassium hydroxide solution until the solution is converted to KHSO3  $(pH \sim 7)$ . In the meantime, a solution of 100 g. of SCl<sub>2</sub> in 1.5 liters of pure petroleum ether is precooled to  $-20^{\circ}$ C. The KHSO, solution is cooled to  $-5^{\circ}$ C. It is then mixed in a 3- to 4-liter stoppered flask with 200-ml. portions of the SCl, solution. The liquid turns yellow in the process and must be decolorized before each new addition by thorough shaking. The temperature of the mixture should not rise above  $+10^{\circ}$ C during the reaction. At the end of the addition the mixture is allowed to stand for some time at  $0^{\circ}C$  in order to complete the separation of the trithionate. The crystalline slurry is suction-filtered, washed with acetone, and dried on clay dishes at room temperature. The yield is 120 g. of approximately 86% K<sub>2</sub>S<sub>3</sub>O<sub>5</sub>, but the product is still contaminated with KCl and sulfur. To recrystallize the salt, it is dissolved in about 350 ml. of H<sub>2</sub>O at 35°C, and the solution is filtered through a heated funnel and rapidly cooled to 0°C (longer heating or higher temperature must be avoided because of the instability of K<sub>2</sub>S<sub>2</sub>O<sub>2</sub>). Completely pure K<sub>2</sub>S<sub>3</sub>O<sub>5</sub> separates. By precipitating the mother liquor with an equal volume of acetone and renewed cooling to 0°C, an additional fraction of the same purity can be obtained. The crystals are filtered with strong suction, washed with acetone, and dried by pressing on clay at room temperature. The yield is 85 g.

### II. REACTION OF SO2 WITH AQUEOUS K2S2O3 SOLUTION

Saturated sulfurous acid (20 ml.) is added to 200 ml. of saturated  $K_2S_2O_3$  solution at 30°C. The flask must be thoroughly cooled with

398

running water. The initial yellow color of the solution disappears after some time. The addition is then repeated, using gaseous  $SO_2$ , until a strong yellow color is apparent. The solution is allowed to stand until the color disappears, and the process is repeated until the yellow color of the solution persists for a fairly long period. After standing for several hours at about 10°C, the precipitated pale yellow crude product is suction-filtered and thoroughly washed with alcohol. To purify the crude compound, it is dissolved in some water and filtered free of suspended S, and pure  $K_2S_3O_6$  is precipitated as shiny needles by mixing the filtrates with an approximately equal quantity of alcohol. The salt is suction-filtered, washed with alcohol and dried on clay at room temperature.

#### **PROPERTIES:**

Colorless crystals with a salty, bitter taste; the pure, dry salt is stable for a fairly long time. Rapidly decomposes into  $SO_2$ , S and  $K_2SO_4$  on heating to  $30-40^{\circ}C$ . d. 2.33.

Readily soluble in water; 100 g. of solution contains 8.14 g. of  $K_2S_3O_6$  at 0°C, 18.43 g. at 20°C; the solution decomposes slowly into  $K_2S_3O_6$  and  $SO_2$ . Insoluble in alcohol.

Crystallizes in  $\tilde{K}$  5<sub>1</sub> structure type.

**REFERENCES:** 

- I. H. Stamm and M. Goehring, Z. anorg. allg. Chem. <u>250</u>, 226 (1942).
- II. F. Martin and L. Metz, Z. anorg. allg. Chem. <u>127</u>, 83 (1923);
   E. H. Riesenfeld, E. Josephy and E. Grünthal, Z. anorg. allg. Chem. <u>126</u>, 281 (1923); H. Hertlein, Z. phys. Chem. <u>19</u>, 287 (1896).

### **Potassium Tetrathionate**

### $K_2S_4O_6$

I.

$$H_2S_4O_6 + 2 KOH = K_2S_4O_6 + 2 H_2O$$
  
112.21 302.46

A two- to three-liter stoppered glass flask is used to saturate 750 ml. of  $H_2O$  with  $SO_2$  at 0°C. Disregarding the appearance of any crystalline  $SO_2 \cdot 6H_2O$ , the solution is reacted at 0°C with 100-ml. portions of a solution of 75 g. of  $S_2Cl_2$  in 500 ml. of

petroleum ether precooled to  $-15^{\circ}$ C. The liquid turns yellow in the process; it must be decolorized before the addition of a new portion by vigorous shaking, following which it is cooled to 0°C. At the end of the addition, the mixture should still have an odor of SO<sub>2</sub>. The petroleum ether is removed in a separatory funnel, and a fast air stream is passed through the aqueous layer for several hours until no further odor of SO<sub>2</sub> can be detected. The solution is then cooled to 0°C and neutralized with an ice-cold solution of 150 g. of KOH in one liter of aqueous alcohol (final pH 6-7). The precipitated  $K_2S_4O_6$ , which still contains about 10% KCl, is suction-filtered and dried on clay at room temperature. The yield is about 165 g. To purify the crude product, it is dissolved in 120 ml. of 70°C water, thoroughly stirred, and, if necessary, reheated rapidly to 60°C. It is then filtered rapidly through a heated funnel; long heating or temperatures above 60°C must be avoided because of the instability of  $K_2S_4O_6$ . On cooling to 0°C, the filtrate yields 120 g. of 100%  $K_2S_4O_5$ . The crystals are filtered by suction, washed with 150 ml. of aqueous alcohol, and dried by pressing on clay at room temperature. By adding the wash alcohol to the mother liquor, a further 20 g. of 99% pure salt can be precipitated.

II.

An aqueous, nearly saturated solution of 39.5 g. of  $K_2S_4O_6 \cdot 1-2/3$  H<sub>2</sub>O (sulfate-free) is added very slowly (drop-by-drop) to an ice-cooled solution of 26 g. of I<sub>2</sub> in a mixture of ethanol and a few milliliters of H<sub>2</sub>O. Very vigorous stirring is needed during the addition. The reaction is instantaneous; the tetrathionate, which is insoluble in ethanol, separates as small crystals. At the end of the addition, the solution is suction-filtered and washed with alcohol until the wash liquor is free of iodine and iodide. To purify the salt, it is redissolved at room temperature in as little water as possible and reprecipitated with alcohol. The precipitate (small, shiny crystals) is completely pure. It is dried by pressing between filter papers and then in a desiccator over concentrated H<sub>2</sub>SO<sub>4</sub>.

PROPERTIES:

Colorless, platelike or prismatic crystals; the pure dry material is stable for a very long time without change but decomposes if  $K_2S_2O_3$  or occluded mother liquor is present, assuming a characteristic odor. On ignition,  $K_2S_4O_5$  decomposes to  $K_2SO_4$ ,  $SO_2$  and S. d 2.29.

Readily soluble in water: 100 g. of solution at  $0^{\circ}C$  contains 12.60 g., at 20°C, 23.18 g. of  $K_2S_4O_6$ ; the solution decomposes slowly to  $K_2S_3O_6$  and  $K_2S_5O_6$ . Insoluble in absolute alcohol.

Monoclinic crystals, space group C-Cc.

**REFERENCES:** 

- I. H. Stamm and M. Goehring, Z. anorg. allg. Chem. <u>250</u>, 226 (1942).
- II. F. Martin and L. Metz, Z. anorg. allg. Chem. <u>127</u>, 83 (1923); A. Sander, Angew. Chem. <u>28</u>, (1915).

#### **Potassium Pentathionate**

### $K_2S_5O_6 \cdot 1.5 H_2O$

Ι.

 $\begin{array}{rl} Na_{2}S_{5}O_{6} \ + \ 2 \ KCH_{3}COO \ = \ K_{2}S_{5}O_{6} \ + \ 2 \ NaCH_{3}COO \\ (1,5 \ H_{2}O) \\ 196.28 \ 361.55 \end{array}$ 

A solution of 51 g. of SCl<sub>2</sub> in 200 ml. of  $CCl_4$  is prepared in a two-liter, stoppered, wide-neck flask and is then cooled to -15°C. At the same time, a solution of 250 g. of Na<sub>2</sub>SO<sub>3</sub> · 5H<sub>2</sub>O in 400 ml. of water is prepared and placed in an ice bath. In a third vessel, 200 ml. of 36% hydrochloric acid is mixed with 200 ml. of water and is also cooled to 0°C. The Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> solution and the hydrochloric acid are then rapidly and simultaneously poured into the SCl<sub>2</sub> solution; the flask is closed and vigorously shaken. The reaction temperature should not exceed 0°C. The mixture becomes colorless almost immediately, or should become so within 20 sec., while the aqueous layer should show only a very slight turbidity due to S. Without delay, 120 ml. of 0.3M FeCl, solution, precooled to 0°C, is then added until the aqueous phase is colored pale yellow. The dark color of the intermediate iron (III) thiosulfate complex is briefly evident and then disappears. The aqueous solution is then separated in a funnel and immediately concentrated at 12 mm. and a bath temperature of 35-40°C to about 170 ml. The residual NaCl is filtered off and the ice-cold concentrate is reacted with ice-cold methanolic potassium hydroxide solution (approximately 20 g. of KOH to 100 ml. of methanol). The alkaline solution is added dropby-drop and the mixture must be continuously and vigorously stirred. The temperature should never rise above +10°C. The brown hydrated iron oxide formed on contact of the two solutions is immediately redissolved by the acid. When the greenish-black hydroxide intermediate begins to separate (pH  $\sim$  3), the addition of

potassium hydroxide is interrupted and the solution is again cooled to 0°C. The separated crystalline slurry is suction-filtered and washed with acetone until the yellowish color disappears. The product is dried on clay at room temperature. The yield is 102 g. of  $85\% K_2 S_5 O_6 \cdot 1.5 H_2 O$ , contaminated with KCl. To recrystallize it, 50 g. of the crude product is added to 100 ml. of 0.5N HCl at 60°C, while the solution (which cools in the process) is rapidly reheated to 50°C and filtered through a heated funnel. The clear solution is placed in a dish set on ice. Star-shaped crystals of 100% pure  $K_2 S_5 O_6 \cdot 1.5 H_2 O$  separate; they are filtered by suction and, after washing with alcohol, are dried on clay. The yield from the entire batch is 46 g. On addition of methanol to the mother liquor, another 13 g. of 80% pure salt is obtained.

### II. REACTION OF THIOSULFATE WITH HYDROCHLORIC ACID IN THE PRESENCE OF ARSENIOUS ACID

A solution of 8-10 g. of As<sub>2</sub>O<sub>3</sub> in 50% sodium hydroxide is prepared and added to a solution of 500 g. of C.P. Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> · 5H<sub>2</sub>O in 600 ml. of water. The reaction vessel is a five-liter flask; the mixture is well stirred and cooled to  $-10^{\circ}$ C (incipient crystallization). Then 800 ml. of concentrated hydrochloric acid (precooled to  $-15^{\circ}$ C) is poured in at once. After thorough mixing, the NaCl precipitate is filtered off on a fritted glass suction filter. The clear filtrate is allowed to stand in a loosely stoppered flask for 3-4 days at  $25^{\circ}$ C. Considerable precipitation of S and As<sub>2</sub>S<sub>2</sub> occurs. The solution is passed through a finely porous filter and is immediately concentrated in vacuum to 200 ml, in a glass rotary evaporator at 38-40°C and 21 mm. (If no evaporator is available, the concentration can also be carried out with a small oil pump at 2-5 mm.; the H<sub>2</sub>O is condensed with an ice-salt or Dry Ice bath; the pump is protected from acid vapors by a CaO drying tower.) The freshly precipitated NaCl is filtered off; the concentrate (d. 1.6) is mixed with 100 ml, of glacial acetic acid and cooled in a tall beaker to -10°C. A thick slurry of fine KC<sub>2</sub>H<sub>2</sub>O<sub>2</sub> crystals is now added in portions while the temperature is kept below  $-2^{\circ}$ C and the mixture is vigorously stirred. (The slurry is prepared by dissolving 80 g. of pure, fused KC<sub>2</sub>H<sub>3</sub>O<sub>2</sub> in 250 ml. of boiling absolute ethanol, cooling with agitation to room temperature and adding - also with vigorous shaking -50 ml. of glacial acetic acid.) The KC<sub>2</sub>H<sub>3</sub>O<sub>2</sub> dissolves because of its fine particle size, and spontaneous separation of K<sub>2</sub>S<sub>5</sub>O<sub>5</sub> · 1.5H<sub>2</sub>O occurs after 30-60 seconds. The crystals are immediately suction-filtered (otherwise partial conversion to  $K_2S_2O_2$  occurs) and washed with a few milliliters of a mixture of two parts of glacial acetic acid and one part of water, then with aqueous alcohol, and finally with absolute ethanol. They are

dried by pressing on clay at room temperature. The yield is 80-100 g. of very pure  $K_2S_5O_6 \cdot 1.5H_2O$ . Addition of a large amount of ethanol to the mother liquor yields another (95% pure) fraction.

PROPERTIES:

Colorless, prismatic or platelike crystals; the pure, dry material is stable for a long time; very unstable on contact with alkali. On heating, the water of crystallization is given off with simultaneous decomposition to  $K_2SO_4$ ,  $SO_2$  and S. d. 2.11.

Readily soluble in water; 100 g. of solution at 0°C contains 15.50 g. and at 20°C, 24.78 g. of  $K_2S_5O_6 \cdot 1.5H_2O$ ; the solution decomposes into  $K_2S_4O_6$  and S. Insoluble in absolute alcohol.

**REFERENCES:** 

- I. M. Goehring and U. Feldmann, Z. anorg. allg. Chem. 257, 223 (1948).
- II. H. Stamm, O. Seipold and M. Goehring, Z. anorg. allg. Chem. <u>247</u>, 277 (1941); see also A. Kurtenacker and W. Fluss, Z. anorg. allg. Chem. <u>210</u>, 125 (1933); F. Foerster and K. Centner, Z. anorg. allg. Chem. <u>157</u>, 45 (1926).

### **Potassium Hexathionate**

### $K_2S_6O_6$

### I. REACTION OF THIOSULFATE WITH NITRITE IN ACID SOLUTION

An aqueous solution of thiosulfate and nitrite is added in one portion and with strong agitation to a mixture of 200 ml. of concentrated hydrochloric acid and 100 ml. of water, precooled to -30 to  $-40^{\circ}$ C, in a three-liter, round-bottom, wide-neck flask. The thiosulfate-nitrate solution is prepared by dissolving 12 g. of C.P. KNO<sub>2</sub> in a filtered solution of 90 g. of technical grade K<sub>2</sub>S<sub>2</sub>O<sub>3</sub> in 90 ml. of water. The mixture first turns dark brown; after a few seconds it becomes dark green and evolves NO vigorously; after 30 seconds the solution turns light green and then, over a period of 2-3 minutes, first yellow and finally a pure white. Until this happens, vigorous agitation is required since otherwise the polythionate solution decomposes with precipitation of sulfur. The nitrogen oxides still present are driven off with a fast N<sub>2</sub> stream and the solution is suction-filtered through glass frit to remove precipitated KC1.

The clear filtrate from two such batches is concentrated at  $25-30^{\circ}$ C and 15-18 mm, to a moderately thick slurry. The crystals

are filtered by suction on fritted glass, washed with aqueous and then absolute alcohol, and dried on clay at room temperature. The yield is 60-70 g. of approximately  $60\% K_2S_6O_6$  contaminated primarily with KCl. To purify it, 50 g. of the crude product is heated in 75 ml. of 2N HCl to about  $80^{\circ}$ C, while the flask contents are vigorously swirled around. The clear, slightly yellow solution which forms is immediately cooled with shaking. The salt precipitate is suction-filtered, washed thoroughly with alcohol, and dried by pressing on clay. The yield from the two batches is 40-44 g. of pale yellow 97.5%  $K_2S_6O_6$ .

II.

 $\begin{array}{rcr} (HCl) \\ S_2Cl_2 \ + \ 2 \operatorname{Na}_2S_2O_3 \ = \ \operatorname{Na}_2S_6O_6 \ + \ 2 \operatorname{Na}Cl \\ (5 \ H_2O) \\ 135.04 \ 496.41 \ 116.91 \end{array}$ 

 $Na_2S_6O_6 + 2 KCH_3COO = K_2S_6O_6 + 2 NaCH_3COO$ 196.28 366.59

The reaction is completely analogous to that for the preparation of  $K_2S_5O_6 \cdot 1.5H_2O$  from SCl<sub>2</sub> and thiosulfate (see above). The details of that procedure can therefore be applied unless noted to the contrary. The two cold solutions (one of 100 g. of Na  $_2S_2O_3 \cdot 5H_2O$  in 150 ml. of water and the other of 80 ml. of 36% hydrochloric acid in 80 ml. of water) are added simultaneously to a solution of 27 g. of S<sub>2</sub>Cl<sub>2</sub> in 100 ml. of CCl<sub>4</sub>, precooled to -15°C in a one-liter wide-neck flask. The mixture is shaken, whereupon the solution becomes colorless. Then about 15 ml. of 0.6M FeCl, solution is added until the aqueous layer turns slightly yellow. It is separated in a funnel, and the aqueous Na<sub>2</sub>S<sub>6</sub>O<sub>6</sub> solution is immediately concentrated at 35°C and 12 mm, to about 50 ml. The NaCl precipitate is filtered off and the concentrate cooled to  $0^{\circ}$ C. Cold methanolic KOH solution is added in drops and with stirring until a pH of 1-2(use indicator paper) is attained. The crystalline slurry is suctionfiltered, washed twice with 40-ml. portions of acetone, and dried on a clay dish. The yield is 42 g. of 81% K<sub>2</sub>S<sub>6</sub>O<sub>6</sub>.

To purify, 20 g. of the crude product is dissolved in 30 ml. of 2N HCl; the clear solution is rapidly heated to  $60^{\circ}$ C and immediately cooled again in ice water, while occasionally swirled around. Filtering by suction, washing, and drying of the crystals are as in method I. The yield from a batch is about 22 g. of 96%  $K_2S_5O_5$ .

PROPERTIES:

Colorless to faintly yellow, copious microcrystalline powder which becomes electrostatically charged with slight friction; stable for a long time in dry air; readily decomposed by alkali.

404

Readily soluble in water (although often slowly); however, the initially clear solution soon decomposes to  $K_2S_5O_6$  and S.

**REFERENCES:** 

- I. H. Stamm, O. Seipold and M. Goehring, Z. anorg. allg. Chem. <u>247</u>, 277 (1941); E. Weitz and F. Achterberg, Ber. dtsch. chem. <u>Ges. 61</u>, 399 (1928).
- II. M. Goehring and U. Feldmann, Z. anorg. allg. Chem. <u>257</u>, 223 (1948).

### Wackenroder Liquid

This is a liquid prepared by passing  $H_2S$  through an aqueous  $SO_2$  solution. In addition to finely divided S and small amounts of  $H_2SO_4$ , it contains mostly higher polythioacids, particularly  $H_2S_5O_6$ . The component distribution varies greatly with the preparative conditions (rate of the  $H_2S$  stream, reaction time, temperature of the solution, etc.).

In the Debus method, a slow stream of pure  $H_2S$  gas is passed for 2-3 hours through 480 ml. of nearly saturated aqueous  $SO_2$ solution held just above 0°C. After the reaction the liquid, which still retains a strong odor of  $SO_2$ , is allowed to stand in the dark for 1-2 days in a closed flask. The  $H_2S$  treatment is then repeated in the same manner. This intermittent  $H_2S$  treatment is continued for about 10-14 days until all of the  $SO_2$  — including that formed during the reaction — is used up; this occurs when the mixture no longer gives off the odor of  $SO_2$  after standing for 10-12 hours at room temperature. The liquid thus obtained is an emulsion. Its thick layers are opaque; thin ones are translucent and red. It can be concentrated on a water bath to d 1.3, in vacuum to d 1.46. No decomposition of the polythioacids occurs in either case, but S precipitates out. The concentrate may be stored in the dark for a long time.

**REFERENCES:** 

H. Debus, Liebigs Ann. Chem. <u>244</u>, 76 (1888). Abegg, Handbuch der anorganischen Chemie, Vol. IV, 1, p. 542.

### **Polythionic Acids**

### $H_2S_xO_3, H_2S_xO_6$

See the original literature for the preparation of these compounds.

**REFERENCES:** 

Max Schmidt, Z. anorg. allg. Chem. <u>289</u>, 141-202 (1957).
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# Nitrosyl Hydrogen Sulfate

### (NO)HSO<sub>4</sub>

 $SO_2 + HNO_3 = (NO)HSO_4$ 64.07 63.02 127.09

Carefully dried SO<sub>2</sub> is introduced into a wash bottle cooled with an ice-salt bath and containing pure, fuming HNO<sub>3</sub> (d 1.60). The reaction is exothermic but the temperature should not rise above  $+5^{\circ}$ C. The SO<sub>2</sub> flow is continued until a thick slurry of (NO)HSO<sub>4</sub> is separated, but some unreacted liquid nitric acid is still present. The crystals are rapidly filtered by suction through a fritted glass filter, washed with glacial acetic acid and CCl<sub>4</sub>, and dried on clay in a desiccator over P<sub>2</sub>O<sub>5</sub>.

SYNONYM:

Nitrosyl sulfuric acid; in industry it is also called lead chamber crystals because of its occurrence as an impurity in the chamber process for manufacture of  $H_2SO_4$ .

PROPERTIES:

White, featherlike to flaky crystals; stable in dry air; sensitive to moisture. M.p.  $73.5^{\circ}C$  (dec.).

Decomposed by water into sulfuric and nitric acids; soluble without decomposition in concentrated  $H_2SO_4$ .

**REFERENCES:** 

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See also G. H. Coleman, G. A. Lillis and G. E. Goheen in: H.S. Booth, Inorg. Syntheses, Vol. I, p. 55, New York-London, 1939.

# Tetrasulfur Tetranitride

 $S_4N_4$ 

 $\begin{array}{rrrrr} 4 \, S &+ \, 6 \, Cl_2 \,+\, 16 \, NH_3 \,=\, S_4 N_4 \,+\, 12 \, NH_4 Cl \\ 128.26 & 425.48 & 272.51 & 184.30 & 641.96 \end{array}$ 

The strongly exothermic process is carried out in an inert organic solvent by reacting  $NH_3$  with a solution (of known concentration) of  $Cl_2$  in  $S_2Cl_2$ .

The reaction is carried out in a six-liter round-bottom flask with a three-hole cork stopper. The stopper holds a large-diameter inlet tube reaching as far down into the flask as possible; the tube has a T connection at the top through which a wire can be pushed to remove any plugs that may form. The inlet tube is connected with an  $NH_3$  cylinder via a flow meter, a pressure release valve, and a long KOH drying tube. The reaction flask is also provided with a power-driven stirrer and a reflux condenser topped with a KOH drying tube.

Four liters of  $CCl_4$  (dried over  $P_2O_5$ ) and 250 ml. of  $S_2Cl_2$  are placed in the flask. This solution is first saturated with  $Cl_2$  at room temperature; then a fast stream of  $NH_3$  (about 50 liters/hour) is passed through with vigorous stirring. The reaction temperature may not exceed 30-50°C; if necessary, the flask should be cooled with ice water.

A thick reddish-brown slurry quickly forms during the reaction. It gradually becomes grayish-green; after 3-4 hours the color becomes lighter. The slurry turns salmon-red after about six hours. At this point, the introduction of NH<sub>3</sub> should be stopped. The precipitate is suction-filtered on a large filter, shaken for 15 minutes with three liters of water, again collected on a frit, and dried on a clay plate. It is then again shaken for one hour with 750 ml. of ether in a one-liter powder bottle to dissolve the byproduct S<sub>7</sub>NH. After filtration and washing with ether, the residue is placed in an extraction tube and treated with dry dioxane at room temperature until the extract has only a slight yellow-orange color. The dioxane solution is carefully concentrated in vacuum. The brownish-red residue is taken up in hot benzene; on cooling, S<sub>4</sub>N<sub>4</sub> crystallizes out in orange-red needles. For further purification, the substance can be sublimed in high vacuum at a bath temperature of about 100°C.

The yield varies; generally, it is about 100 g.

It must be kept in mind during the entire procedure that  $S_4N_4$  is susceptible to explosive decomposition induced by shock or temperatures above 100°C.

PROPERTIES:

Light yellow-orange solid at ordinary temperature; becomes light yellow at  $-30^{\circ}$ C; on heating to  $100^{\circ}$ C, orange-red. M.p.  $178^{\circ}$ C, b.p. about  $185^{\circ}$ C; decomposes explosively on further heating; d 2.22.

Insoluble in water; readily soluble in many organic solvents such as benzene,  $CS_2$  and dioxane; only moderately soluble in alcohol and ether.

Monoclinic crystals, type  $C_{zh}$ ; space group  $P2_1/n$ .
**REFERENCE:** 

M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 144, 3, 17.

## **Tetrasulfur Dinitride**

 $S_4N_2$ 

 $S_4N_4 + 4S = 2S_4N_2$ 184.3 128.3 312.6

The reaction vessel is a 500-to 750-ml. stirred autoclave which can be heated to  $110^{\circ}$ C.

A mixture of 24 g. of  $S_4N_4$  and 50 g. of S is dissolved or suspended in 380 ml. of pure  $CS_2$  and heated in the autoclave for two hours at 110°C. The mixture is then cooled as rapidly as possible. If a great deal of thiocyanogen polymer forms during the reaction, it is filtered off. The residue is thoroughly washed with  $CS_2$  and the wash liquor is combined with the filtrate. The carbon disulfide is then evaporated in vacuum. The red evaporation residue is distilled in high vacuum at a bath temperature of 60-65°C. Darkred crystals separate in a trap cooled with Dry Ice-methanol mixture.

The yield is about 4 g. The autoclave should not be cleaned between runs since the yield is good only if the walls are contaminated with material from a previous  $S_4N_2$  preparation.

PROPERTIES:

M.p. 23°C; quite unstable; decomposes within a few hours even at 0°C; decomposes explosively to S and N at 100°C; soluble in many organic solvents; insoluble in water; hydrolyzes slowly with water; diamagnetic.

REFERENCE:

M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 22, 147.

# Disulfur Dinitride S<sub>2</sub>N<sub>2</sub>

 $S_4N_4 = 2 S_2N_2$ 184.3 184.3

The compound is prepared by thermal degradation of  $S_4N_4$ . The reactor is a quartz tube, about 320 mm. long. The diameter of the lower section of the tube is 11 mm., while that of the top half is approximately twice that. As shown in Fig. 167, the upper part of the tube is connected through two condensation traps to a high-vacuum pump; a water-cooled glass finger extends into the upper section of the reactor. All connections are ground glass joints. The narrower, lower section of the reactor is surrounded by two electric furnaces, each heating about half of the section. The lower furnace heats the contents to about  $80^{\circ}$ C, the upper one to about  $300^{\circ}$ C.



Fig. 167. Preparation of disulfur dinitride.

A small amount (1-2 g.) of  $S_4N_4$  is placed in the reactor. A 7-cm.-long part of the section heated by the upper furnace is filled with tightly compressed silver wool. The apparatus is evacuated to 0.005 mm. The first condensation trap is then cooled with Dry Ice-methanol and the second trap with liquid nitrogen. The upper furnace is heated to 300°C and only then is the lower one switched on.

The cold finger soon becomes coated with a blue film, and colorless to faintly yellow crystals form in the inlet tube to the first trap. After 6-8 hours, the thermal degradation of  $S_4N_4$  is complete. The furnaces and the trap coolants are then removed and the apparatus is flushed with dry air or dry  $N_2$ .

The light-gray crystalline coating in the first trap (which is reddish, with a blue rim, where it extends beyond the cooling zone) is extracted several times with 10-ml. portions of dry ether until only a few dark-blue or shiny metallic crystals remain. The first extract is a deep red because of the byproduct  $S_4N_2$ ; since  $S_4N_2$  dissolves readily, the color of the other extracts is lighter. The combined ether solution is filtered and placed in a conical ground glass flask (which narrows to a point at the bottom) provided with an adapter permitting reverse filtration with exclusion of moisture. The flask is then cooled to  $-80^{\circ}$ C in a Dry Icemethanol bath; the white  $S_2N_2$  crystals precipitate. These are separated from the ether by reverse filtration.

To purify the  $S_2N_2$  it can be sublimed at room temperature in high vacuum. Beautiful, large, colorless crystals are obtained. The yield, prior to sublimation, is 80%.

It cannot be overemphasized that the preparation and further manipulation of  $S_2N_2$  must be done very cautiously. Significant polymerization occurs within a short time even at room temperature. The material detonates violently at 30°C, or when under high mechanical pressure.

## PROPERTIES:

Well-formed, colorless crystals; very volatile; unpleasant iodinelike odor; stable only at low temperature; becomes dark after a short exposure to  $20^{\circ}$ C; sublimes at  $10^{-2}$  mm. even at room temperature; polymerizes readily to  $(SN)_X$ ; in the presence of traces of moisture, about 67% of the S<sub>2</sub>N<sub>2</sub> polymerizes to (SN)<sub>x</sub>, while 33% dimerizes to  $S_4N_4$ ; detoned by shock, friction and temperature above 30°C; soluble in alcohols, yielding yellowish red solutions; readily forms colorless solutions in benzene, ether, carbon tetrachloride, acetone, tetrahydrofuran, dioxane; in the absence of moisture, the colorless solutions are more stable than the solvent-free substance (however, the addition of traces of alkali metals, some NaOH, KCN or Na<sub>2</sub>CO<sub>3</sub> causes instantaneous and complete dimerization); crystals are not wetted by water and acids (for this reason, hydrolysis with these solvents occurs very slowly); vigorous reaction with alkali solutions; dissolves rapidly in 2 N NaOH, giving a yellow solution, with larger crystals becoming black and detonating, giving off a pungent gas.

#### **REFERENCE:**

M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 18, 145.

## Sulfur Nitride Tetrahydride

 $S_4(NH)_4$ 

A solution of 10 g. of  $S_4N_4$  in 300 ml. of dry benzene is heated to  $80^{\circ}C$  in a two-liter flask. A solution of 35 g. of  $SnCl_2 \cdot 2H_2O$  in 80 ml. of methanol containing about 5% water is added all at once. The solution starts to boil and becomes colorless. The precipitate formed is suction-filtered and washed with cold 2N HCl until no Sn remains. It is then washed with alcohol and ether. Further purification is by recrystallization from methanol.

The yield is about 6 g.

PROPERTIES:

Colorless solid; small, shiny crystals of  $C_{2h}$  type; reddens on heating to 80-100°C; m.p. 152°C (provided it is heated very rapidly; otherwise decomposition occurs); not wetted and not dissolved by water; readily soluble in pyridine, slightly soluble in hot acetone and hot alcohol, very slightly soluble in other organic solvents; diamagnetic.

**REFERENCE:** 

M. Goehring, Scientia Chimica <u>9</u>, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 28, 147.

## Heptasulfur Imide

#### S7NH

A two-liter, three-neck flask, equipped with a stirrer, a gas inlet tube reaching to the bottom, and a gas outlet tube connected to a KOH drying tube, serves as the reaction vessel.

The flask is filled with one liter of dimethylformamide and cooled with an ice-salt mixture, and a fast stream of  $NH_3$  is passed through with vigorous stirring. When the solution is saturated with  $NH_3$  and the temperature has dropped to  $-5^{\circ}C$ , the gas outlet tube

is briefly removed and 5 ml. of S<sub>2</sub>Cl<sub>2</sub> is rapidly injected from a pipette without interrupting the inflow of NH<sub>2</sub>. After the addition of S<sub>2</sub>Cl<sub>2</sub>, the temperature of the reaction mixture increases somewhat; after renewed cooling to  $-5^{\circ}$ C, another 5 ml. of S<sub>2</sub>Cl<sub>2</sub> is injected into the flask. A total of 100 ml. of S<sub>2</sub>Cl<sub>2</sub> is added in this manner. Ammonia passage is continued for 15 min., after which stirring is interrupted. The mixture is left to stand for one hour and is then poured into three liters of 1% HCl (precooled to  $0^{\circ}$ C). Some ice is added. The mixture is stirred, neutralized with 10% HCl, and left to stand for 2-3 hours to settle the reaction products (S and S<sub>7</sub>NH). The supernatant is then decanted; the solid products are washed with water on a filter and dried in a vacuum desiccator over CaCl<sub>2</sub>. The crude product is shaken with 250 ml. of tetrahydrofuran for one half hour. It is then filtered through a fluted filter paper, and the tetrahydrofuran is removed in aspirator vacuum. The residue is recrystallized from hot methanol. By concentrating the mother liquor, further heptasulfur imide can be obtained. The yield is 16-20 g.

### PROPERTIES:

M.p. 113.5°C; crystallizes in space group  $D_{2h}^{16}$ ; d 2.01. Not wetted or dissolved by water; readily soluble in organic solvents; characteristic blue-violet color with acetone when alcoholic alkali hydroxide is added.

REFERENCES:

- M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, p. 55.
- M. Becke-Goehring, H. Jenne and E. Fluck, Chem. Ber. <u>91</u>, 1947 (1958).

## a-Sulfanuric Chloride

[OS(N)Cl]<sub>3</sub>

$3 SO_2 Cl_2$	+ 3 SOCl <sub>2</sub>	$+ 12 \text{ NH}_{3} =$	$[OS(N)Cl]_3$	$+ 3 SO_2$	$+ 9 \text{ NH}_4 \text{Cl}$
405.0	357.0	204.4	292.8	192.2	481 <b>.5</b>

A mixture of sulfuryl chloride and thionyl chloride (100 ml., mole ratio 2 : 1) is diluted with 100 ml. of low-boiling petroleum ether in a 500-ml., three-neck, ground glass flask equipped with a stirrer, a gas inlet and a drying tube. The flask is cooled in a Dry Ice-methanol bath, and a rapid stream of dry  $NH_3$  is passed through for 1.5 hours, while the mixture is continuously stirred.

A yellow slurry is formed. The solvent is removed in vacuum at a bath temperature of about  $50^{\circ}$ C; the residue is washed with cold water, immediately filtered and dried in a vacuum desiccator. From this water-insoluble reaction mixture,  $\alpha$ -sulfanuric chloride is obtained by sublimation at a bath temperature of  $60-80^{\circ}$ C and a pressure of 0.05 mm.

The yield is about 1 g.

PROPERTIES:

White, crystalline substance; not wetted by water; soluble in organic solvents; forms esters with alcohols; hydrolyzes slowly with water; m.p.  $144.5^{\circ}$ C.

REFERENCE:

M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 96, 158.

## Trisulfur Dinitrogen Dioxide

## $S_3N_2O_2$

I.

 $\begin{array}{c} S_4N_4 \ + \ 4 \ \text{SOCl}_2 \ + \ 4 \ \text{SO}_2 \ = \ 2 \ S_3N_2O_2 \ + \ 4 \ \text{SO}_2Cl_2 \ + \ 2 \ \text{S} \\ 184.3 \ \ 475.9 \ \ 256.2 \ \ 312.4 \ \ 539.9 \ \ 64.1 \end{array}$ 

The reaction vessel is a 100-ml., two-neck, ground glass flask with reflux condenser and gas inlet tube.

Approximately 1 g. of  $S_4N_4$  is reacted with a mixture of equal volumes of  $C_6H_6$  and SOCl<sub>2</sub>. The quantity of the  $C_6H_6$ -SOCl<sub>2</sub> mixture should be such that a portion of the  $S_4N_4$  remains undissolved. Dry SO<sub>2</sub> is then introduced and the flask is heated at 70-75°C for two hours. The solution turns red-brown. The solvent is then evaporated in vacuum at room temperature and the  $S_3N_2O_2$  is sublimed from the mixture in high vacuum at about 40°C to form large yellow crystals.

The yield is about 0.8 g.

II. Approximately 80 g. of distilled SOCl<sub>2</sub> is diluted with 80 g. of dry petroleum ether and cooled to  $-80^{\circ}$ C. A rapid stream of dry NH<sub>3</sub> is passed through this solution until the mixture becomes a thick yellow slurry. The reaction must be protected from moisture. The solvent is distilled off in vacuum at room temperature and the dry residue is then transferred into a sublimation vessel; the

product sublimes at a bath temperature of about 40°C and 0.01 mm. The crude product may be recrystallized from dry benzene.

The yield is about 6 g.

PROPERTIES:

Well-formed, pale yellow crystals (rhombic, pseudotetragonal); soluble in organic solvents (e.g., benzene, nitrobenzene, heptane, petroleum ether, alcohols; these solutions are stable if the solvent is completely dry); hydrolysis occurs only on prolonged contact with water; immediate hydrolysis in alkaline solution; completely stable when stored in dry air (or dry N<sub>2</sub>, dry SO<sub>2</sub>) at room temperature; becomes red when heated to  $80^{\circ}$ C; m.p.  $100.7^{\circ}$ C (without decomposition); boils on further heating, evolving a yellow vapor which ignites spontaneously in air at about  $300^{\circ}$ C. When decomposed by moisture, the yellow crystals first turn red, then black and finally white (the odor of SO<sub>2</sub> is apparent).

**REFERENCE:** 

M. Goehring, Scientia Chimica <u>9</u>, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 156, 85.

## Trisulfur Dinitrogen Pentoxide

 $S_3N_2O_5$ 

The reactor is a 4-cm.-diameter tube with a capacity of 200 ml. A water-cooled cold finger, extending through about 4/5 of the tube length, is inserted through a ground glass connection. A flask filled with oleum, a vacuum line, and a pressure release valve (protected with an  $H_2SO_4$  drying tube) are connected to the reaction tube via stopcocks.

A 5-g. quantity of  $S_4N_4$  is placed in the tube. The cold finger is then cooled with running water and the tube is evacuated to about 15 mm. The stopcock to the flask containing the oleum is now opened; the flask is evacuated and heated until SO<sub>3</sub> distills out. It condenses on the cold finger and then drips onto the sulfur nitride below. Immediately after the start of the reaction, the reactor is cooled with ice water. The  $S_4N_4$  first turns blackbrown and SO<sub>2</sub> is evolved. Later, when the substance reverts to light yellow and is covered with some liquid SO<sub>3</sub>, the stopcock to the oleum flask is closed and the stopcock to the pressure release valve is opened so that the  $SO_2$  may escape. The cooling bath is now removed and the mixture allowed to stand at room temperature for about six hours. The excess  $SO_3$  is distilled back into the oleum flask by opening the stopcock to this flask, evacuating the reaction vessel to 15 mm., cooling the oleum flask to 0°C, and heating the reaction tube to 30°C. If the reactor is then heated briefly to 60°C, even the last traces of  $SO_3$  can be removed. Finally, the stopcock to the oleum flask is closed again, the apparatus is evacuated to 1 mm., and the reactor is heated to 70-80°C. Most of the  $S_3N_2O_5$  product sublimes onto the cold finger.

The yield is about 5 g.

**PROPERTIES:** 

Well-formed, almost colorless crystals, which can be sublimed in vacuum; completely stable when stored dry; readily soluble in organic solvents (e.g., nitrobenzene) without decomposition; vigorous reaction with traces of moisture. The compound is characterized most easily by its powder pattern.

**REFERENCE**;

M. Goehring, Scientia Chimica 9, Ergebnisse und Probleme der Chemie der Schwefelstickstoffverbindungen [Data and Problems in the Chemistry of Sulfur-Nitrogen Compounds], Berlin, 1957, pp. 156, 85.

## Selenium

Se

#### VERY PURE SELENIUM

Commercial selenium frequently contains some sulfur as well as small amounts of tellurium and iron. To remove these, it is oxidized to  $SeO_2$ ; the latter is purified by repeated sublimation and is then reduced again to elemental selenium.

The oxidation of Se to  $SeO_2$  should be carried out according to the procedure given under  $SeO_2$ , with particular attention to the purity of all reagents used. The subsequent resublimation is also carried out in the manner described. The final sublimate, consisting of loose, colorless crystals, is dissolved in water, refiltered (if needed), and then reduced in a stirred flask placed on the water bath by the gradual addition of 10% hydrazinium hydroxide solution. The Se precipitates as a red powder which, when heated, soon coagulates to form a grayish-black precipitate. An excess of the reducing agent should be avoided as it dissolves Se, forming red polyselenides. (These may be decomposed by the addition of hydrochloric acid.) The Se is filtered off and washed several times with hot water until the filtrate ceases to give a blue color on addition of KI-starch solution. The entire purification procedure is then repeated. The dark powder obtained after the second reduction is thoroughly washed, dried at  $170^{\circ}$ C, and, if desired, distilled in a stream of N<sub>2</sub> to remove the last traces of adsorbed moisture (a quartz apparatus should be used).

**REFERENCES:** 

- J. Jannek and J. Meyer, Z. anorg. allg. Chem. 83, 51 (1913).
- O. Hönnigschmid and W. Kapfenberger, Z. anorg. allg. Chem. 212, 198 (1933).

## MODIFICATIONS OF SELENIUM

a) AMORPHOUS SELENIUM

Amorphous selenium is formed by reduction of Se compounds at moderate temperatures. Depending on the state of aggregation, it may be red to black in color.

To prepare amorphous red Se,  $SO_2$  is passed at 15 to  $20^{\circ}C$  through an aqueous  $H_2SeO_3$  solution strongly acidified with hydrochloric acid. The finely powdered Se precipitate is carefully washed free of Cl<sup>-</sup> and  $SO_4^{2^-}$  ions and dried in a vacuum desiccator over CaCl<sub>2</sub>.

Dark, amorphous Se is obtained by treating red Se with boiling water. It is also formed by reduction of heated selenous acid with hydrazinium hydroxide (see Very Pure Selenium above). The grayish-black powder is thoroughly washed with warm water and dried in vacuum over CaCl<sub>2</sub>.

Both forms of amorphous Se are stable at room temperature.

#### b) AMORPHOUS, VITREOUS SELENIUM

Vitreous Se is formed on rapid cooling of molten Se.

Solid selenium of any available modification is melted and poured in a thin stream into coldwater. Strands of brittle, vitreous Se are obtained. Thin layers of this material are translucent and red, while thick layers are grayish black. The substance can be stored for a long time at room temperature.

c) MONOCLINIC a -SELENIUM AND  $\beta$  -SELENIUM

The two monoclinic modifications are formed together when amorphous Se is treated with  $CS_2$ .

A few grams of amorphous red Se are refluxed for two hours in one liter of CS<sub>2</sub>. The orange, slightly green-tinted solution is allowed to evaporate slowly at room temperature in a vessel protected from dust. Large red crystals of  $\alpha$ -Se along with smaller, dark prisms of  $\beta$ -Se are formed. They can be separated by sorting under the microscope. Both forms are quite stable at room temperature.

## d) HEXAGONAL SELENIUM

Metallic, hexagonal selenium is formed when any one of the other selenium modifications is heated above  $130^{\circ}C$ .

Monoclinic Se, vitreous Se or powdered amorphous Se is vacuum heated for some time (10 hours or more) at 200°C until complete conversion to the metallic form has occurred. The product is lead gray, finely crystalline and granular.

A highly ordered Se solid, in which no lattice imperfections can be detected by x ray analysis, is obtained only after annealing for several days at 200 to  $218^{\circ}$ C. The crystallization of Se can be catalyzed by various substances.

For methods of preparing single crystals up to 10 mm. long and 0.5 mm. in diameter, see the cited literature.

## e) COLLOIDAL SELENIUM

Stable, essentially monodispersed Se sols are obtained by reduction of selenous acid with hydrazinium hydroxide. A mixture of 90 ml. of distilled  $H_2O$  and 5 ml. of  $1.5M N_2H_BOH$  solution is heated, and 4 ml. of pure  $H_2SeO_3$  solution (0.1M) is added when the temperature reaches  $100^{\circ}C$ . When the mixture turns dark yellow, 1 ml. of the same  $H_2SeO_3$  solution is added. The flame is then removed; the solution is cooled for about 10 minutes and diluted to 400 ml. with distilled  $H_2O$ . When stored in the dark, the sol is stable for a fairly long period. On the other hand, it flocculates quantitatively in a short time on exposure to direct sunlight. The particle count is  $3-4 \cdot 10^{10}$ /ml. If the 5 ml. of  $H_2SeO_3$  solution is added all at once to a vigorously boiling solution of reducing agent, a greater number of particles ( $10-12 \cdot 10^{10}$ /ml., diameter about 75 mµ) is obtained. The sols may be freed of the electrolyte by dialyzing.

#### **REFERENCES:**

- a) S. S. Bhatnagar, M. R. Verma and M. Anwar-ul-Hag, Kolloid-Z. <u>78</u>, 9 (1937); V. Lenher, J. Amer. Chem. Soc. <u>20</u>, 565 (1898).
- b) K. Neumann and E. Lichtenberg, Z. phys. Chem. (A) 184, 89 (1939); V. Regnault, Ann. Chim. Phys. [3] 46, 284 (1856).

- c) W. Muthmann, Z. Kristallogr. <u>17</u>, 336 (1890); F. Halla, F. X. Bosch and E. Mehl, Z. phys. Chem. (B) <u>11</u>, 455 (1931); H. P. Klug, Z. Kristallogr. <u>88</u>, 130 (1934).
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- e) H. R. Kruyt and E. A. van Arkel, Rec. Trav. Chim. Pays Bas 39, 656 (1929); Kolloid-Z. 32, 29 (1923); F. B. Gribnau, Kolloid-Z. 82, 15 (1938).

## Hydrogen Selenide

## H<sub>2</sub>Se

Hydrogen selenide is even more poisonous than  $H_2S$ . It very strongly attacks the mucous membranes of the eyes, nose and throat ("selenium fever"). The preparation must therefore be carried out under a very good hood, using a carefully sealed apparatus with the outlets directly connected to the stack.

I.  $Al_2Se_3 + 6H_2O = 3H_2Se + 2Al(OH)_3$ 290.82 108.10 242.93

The reactor is a dry 500-ml. ground glass flask equipped with a separatory funnel and gas inlet and outlet tubes. The outlet tube is connected to the drying and condensation apparatus; the latter consists of two U tubes, containing CaCl<sub>2</sub> and P<sub>2</sub>O<sub>5</sub>-glass wool, as well as two glass condensation traps cooled to  $-78^{\circ}$ C. The flask is filled with pure dry Al<sub>2</sub>Se<sub>3</sub> (for preparation see section on Aluminum) and the apparatus is flushed for 15-20 minutes with oxygen-free, dry N<sub>2</sub> until all air is displaced. While a slow stream of N<sub>2</sub> is passed through, freshly distilled cold water is added slowly from the separatory funnel. The addition must be regulated so that the reaction is not too violent and the flask heats up only very slightly. A little dilute hydrochloric acid is added toward the end, when the gas evolution subsides. The condensate collected in the traps is 100% H<sub>2</sub>Se<sub>3</sub> were pure. The yield is about 85%.

Hydrogen selenide can be stored in liquid form at low temperatures or as a vapor in sealed glass flasks.

II. 
$$H_2 + S_E = H_2S_E$$
  
2.02 78.96 80.98

A mixture of oxygen-free, dry  $H_2$  and pure selenium vapor is passed through a combustion tube filled with pumice fragments and heated to  $350-400^{\circ}$ C. For the arrangement of the apparatus and the procedure, see the analogous synthesis of H<sub>2</sub>S (p. 344) as well as the original literature. After passing through traps cooled to  $-20^{\circ}$ C and  $-40^{\circ}$ C, the H<sub>2</sub>S is frozen out with liquid nitrogen; after another distillation in high vacuum, it is very pure.

**PROPERTIES:** 

Colorless gas with an unpleasant odor "reminiscent of rotten radishes"; very poisonous. Decomposed by dust, rubber and, in the presence of moisture, also by atmospheric oxygen, with separation of selenium. For this reason, rubber connections are to be avoided. M.p.  $-65^{\circ}$ C, b.p.  $-42^{\circ}$ C, crit. t.  $+141^{\circ}$ C, crit. p. 91 atm. d. (liq.  $-42^{\circ}$ C) 2.12, d. (solid  $-170^{\circ}$ C) 2.45; weight per liter 3.6643 g.

More soluble in water than  $H_2S$ ; solubility at 760 mm. in 1 vol. part  $H_2O$  (4°C) 3.77; (13.2°C) 3.31; (22.5°C) 2.70 vol. parts  $H_2Se$ . In air, the initially colorless solution rapidly becomes cloudy, with separation of red Se.

The Se atoms in crystalline  $H_2$ Se form a cubic face-centered lattice; probably structure type C1 or C2.

**REFERENCES:** 

- I. G. R. Waitkins and R. Shutt in: W. C. Fernelius, Inorg. Syntheses, Vol. II, p. 183, New York-London, 1946; see also L. Moser and E. Doctor, Z. anorg. allg. Chem. <u>118</u>, 284 (1921).
- II. A. Klemenc, Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], Leipzig, 1938, p. 184;
  W. Hempel and M. G. Weber, Z. anorg. allg. Chem. 77, 48 (1912); M. G. Weber, Thesis, Dresden, 1910; H. J. Backer, Rec. Trav. Chim. Pays Bas <u>62</u>, 580 (1943).

# Sodium Hydrogen Selenide NaHSe

Because of the instability of NaHSe and  $H_2Se$  in moist air, the preparation must be carried out in the closed apparatus shown in Fig. 168.

Oxygen-free, dry  $N_2$  is introduced at a, and the apparatus is carefully purged until all air is displaced. Then a small tube, containing highly pure, vacuum-distilled Na (see section on Alkali Metals), the tip of which has been broken just prior to insertion, is pushed with the open end down into attachment C, and the latter is rapidly closed off with a rubber stopper. After evacuating through 2 with stopcocks 1, 3 and 4 closed, the metal is melted in high vacuum and allowed to flow into reaction vessel A. Capillary  $k_1$  is then fused to form a seal. Section A is then cooled with an ice-salt mixture, and carefully dehydrated alcohol is added from D in small portions until an alcoholic ethoxide solution is formed. When the



Fig. 168. Preparation of sodium hydrogen selenide.

reaction is complete, A is warmed again to room temperature. Then a mixture of pure dry  $H_2$  and  $H_2$ Se is allowed to enter at a with the three-way stopcock 1 turned so that the gas escapes at d and thus flushes the stopcock holes free of air; only then is it introduced into the apparatus by turning stopcock 1. The solution heats up considerably during the reaction. At first the gas is allowed to escape at 3; when section B is sufficiently filled with  $H_2$ Se, the gas is allowed to escape at 2. As soon as the reaction subsides, the  $H_2$  inflow is stopped and pure  $H_2$ Se is introduced. The end of the reaction is recognized by the absence of vapor mists and cooling of vessel A. Stopcock 2 is then closed and stopcock 3 opened, the apparatus is tilted, and the NaHSe solution is forced to flow toward B under hydrogen pressure. Capillary  $k_2$  is then fused shut. The alcohol is evacuated with an aspirator through c. Finally, high vacuum is applied, and the remaining fine, white crystals of NaHSe are dried at about 50°C. The product is then analytically pure.

PROPERTIES:

White crystals; very unstable in moist air, separating Se and forming polyselenides.

Rhombohedrally distorted NaCl structure.

REFERENCE:

W. Teichert and W. Klemm, Z. anorg. allg. Chem. 243, 86 (1939).

## Sodium Selenide, Potassium Selenide

Na<sub>2</sub>Se, K<sub>2</sub>Se

The reaction is carried out in liquid  $NH_3$ . For apparatus and procedure, see  $K_2S$  (p. 360). The directions given there may be applied without change. The white  $Na_2Se$  or  $K_2Se$  is analytically pure.

PROPERTIES:

White, fine, hygroscopic crystals; rapidly turns red in moist air because of Se separation and formation of polyselenides.

Na<sub>2</sub>Se:  $m.p. > 875^{\circ}C$ , d 2.58.

 $K_2$ Se: On heating, initially brown, then black. d 2.29 (by x ray). Both compounds are soluble in water; red Se separates rapidly on contact with air.

Crystallizes in Cl structure type.

REFERENCE:

W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. 241, 281 (1939).

## Sodium Diselenide

 $Na_2Se_2$ 

 $Na_2Se + Se = Na_2Se_2$ 124.95 78.96 203.91

About 2.5 g. of  $Na_2Se$  and the stoichiometric amount of Se are placed in a small glass tube in a vigorous oxygen-free stream of

dry N<sub>2</sub>. The tube is sealed off in high vacuum and then heated in an electric furnace to  $500^{\circ}$ C until a homogeneous melt is formed (about 45 min.). The gray-black, very hard mass that forms on cooling is Na<sub>2</sub>Se<sub>2</sub>.

PROPERTIES:

Dull gray-black, microcrystalline substance; very hygroscopic; decomposes in moist air, turning red.

M.p. about 495°C.

Readily soluble in water, forming a red solution which decomposes rapidly on contact with air, separating selenium.

**REFERENCES:** 

- W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. 241, 281 (1939).
- W. Klemm and H. Sodomann, Z. anorg. allg. Chem. 225, 273 (1935).

## Diselenium Dichloride

Se<sub>2</sub>Cl<sub>2</sub>

Ι.

 $3 \text{ Se} + \text{ SeO}_2 + 4 \text{ HCl} = 2 \text{ Se}_2 \text{Cl}_2 + 2 \text{ H}_2 \text{O}$ 236.88 110.96 145.86 457.67

A solution of 115 g. of freshly sublimed SeO<sub>2</sub> in 500 ml. of concentrated hydrochloric acid (36-37%) is prepared in a wide-neck flask, and 235 g. of pure Se is added. Then concentrated  $H_2SO_4$  is added in drops from a separatory funnel until HCl evolution ceases. The mixture is cooled. The Se<sub>2</sub>Cl<sub>2</sub>, which settles at the bottom as a red layer, can then be removed in the separatory funnel. To purify the product, it is either washed several times with concentrated  $H_2SO_4$ , or it is dissolved in fuming sulfuric acid and reprecipitated with HCl. Any adhering  $H_2SO_4$  is removed by treating with anhydrous BaCl<sub>2</sub> and subsequent filtering through fritted glass. The filtration must be protected from moisture. The yield of redbrown Se<sub>2</sub>Cl<sub>2</sub> is about 90%.

II.  $2 \text{ Se} + 2 \text{ SO}_3 + 3 \text{ HCl} = \text{Se}_2 \text{Cl}_2 + \text{H}_2 \text{SO}_3 + \text{SO}_2(\text{OH}) \text{Cl}$ 157.92 160.13 109.40 228.83 82.08 116.54

A distillation flask is filled with 300 g. of 30% oleum, and 100 g. of finely powdered Se is added with constant stirring. The flask

is equipped with a gas inlet tube extending to the bottom and is connected to a downward condenser, connected to a receiver. From the receiver, a gas outlet tube leads (through an  $H_2SO_4$  wash bottle) directly to the stack of the hood. A fast stream of dry HCl (p. 280) is bubbled through the reaction mixture, which is carefully heated until Se just begins to volatilize. After a short time, crude  $Se_2Cl_2$ begins to distill into the receiver at a uniform rate. The distillation rate is adjusted so that about 130 g. distills in two hours. The crude product is shaken several times with small amounts of fuming sulfuric acid until it no longer turns green (SeSO<sub>3</sub>), then allowed to stand for some time over anhydrous BaCl<sub>2</sub>, and finally filtered through a fritted glass filter protected from access of moisture.

Stored in sealed glass ampoules.

SYNONYMS:

Older names: selenium monochloride and selenous chloride.

PROPERTIES:

Dark red, oily liquid with a brown tint; odor similar to  $S_2Cl_2$ ; sensitive to moisture. M.p.  $-85^{\circ}C$ , b.p. (733 mm.)  $127^{\circ}C$  with partial decomposition into Se and SeCl<sub>4</sub>; cannot be distilled without decomposing, even at reduced pressure.  $d_4^{25}$  2.77.

Slowly hydrolyzed by water, forming H<sub>2</sub>SeO<sub>3</sub>, HCl and Se. Dissolves without decomposition in CS<sub>2</sub> and CHCl<sub>3</sub>.

**REFERENCES:** 

- I. V. Lenher and C. H. Kao, J. Amer. Chem. Soc. <u>47</u>, 772 (1925); 48, 1550 (1926).
- II. F. H. Heath and W. L. Semon, Ind. Eng. Chem. 12, 1100 (1920).

## Selenium Tetrachloride

SeCl<sub>4</sub>

 $\begin{array}{rrrr} Se \ + \ 2 \ Cl_2 \ = \ SeCl_4 \\ 78.96 \ 141.83 \ 220.79 \end{array}$ 

I. DIRECT CHLORINATION OF SOLID Se:

A 1- to 15-cm. section of a 1.5-m.-long combustion tube is filled with pure granulated Se (free of Te). The filled section starts at a distance of 10 cm. from one end. The tube is closed at both ends

with one-hole rubber stoppers holding gas inlet and outlet tubes. The inlet is at the end close to the Se layer. The outlet tube is connected, through a  $CaCl_2$  drying tube, to the hood stack. The middle and rear sections of the tube are cooled with strips of wet filter paper, and pure, dry chlorine gas is introduced at room temperature. The reaction begins immediately, with absorption of  $Cl_2$  and heating of the selenium layer. Toward the end of the reaction, the selenium must be heated somewhat with a multiple-flame burner. The SeCl<sub>4</sub> deposits in the colder section of the tube. It is then sublimed several times (in the direction of the tube rear) in a fast stream of  $Cl_2$  in order to remove small amounts of SeOCl<sub>2</sub>. The pure product is loosened from the glass wall by light tapping, with heating if necessary. It is then quickly shaken out of the reactor and sealed in ampoules. The yield is about 90%.

A different apparatus operating on the same principle is described by H. G. Nowak and J. F. Suttle in: T. Moeller, Inorganic Syntheses, Vol. V, p. 125, New York-Toronto-London, 1957.

## II. CHLORINATION OF Se IN AN INERT SOLVENT:

Pure Se is suspended in  $CCl_4$  in a round-bottom flask provided with a gas inlet and outlet. Dry  $Cl_2$  is then introduced. The Se soon dissolves and the solution turns brown (formation of  $Se_2Cl_2$ ); after some time,  $SeCl_4$  separates as a yellow-white powder. It is rapidly filtered by suction (the filtration must be protected from moisture) on a fritted glass filter, washed with a small amount of  $CCl_4$ , and dried in a vacuum desiccator over silica gel.

#### **PROPERTIES:**

White to faintly yellow crystalline substance; decomposes in moist air. On heating, sublimes without melting; the change in the color of the vapor from yellow to red with increasing temperature indicates increasing dissociation; subl. p. about  $196^{\circ}$ C. Melts in a closed tube at about  $305^{\circ}$ C to form a dark red liquid. d 3.80.

With water, it hydrolyzes exothermically to form  $H_2SeO_3$  and HCl.

**REFERENCES:** 

- I. G. Wagner, Anorganisch-präparatives Praktikum [Inorganic Preparative Manual], Vienna, 1947, p. 114; F. Clausnizer, Liebigs Ann. Chem. <u>196</u>, 265 (1879).
- II. V. Lenher, J. Amer. Chem. Soc. 42, 2498 (1920); H. C. Bell and C. S. Gibson, J. Chem. Soc. (London) <u>127</u>, 1877 (1925).

## Hexachloroselenium Salts

The stock solution for the preparations is prepared by dissolving 5.5 g. of  $SeO_2$  in hydrochloric acid; alternatively, 4 g. of finely powdered gray Se is added to 50 ml. of concentrated hydrochloric acid, and  $Cl_2$  is passed through with frequent swirling or stirring until the initially dark brown sludge of  $SeCl_2$  is completely dissolved and the solution is brownish-yellow.

(NH<sub>4</sub>)<sub>2</sub>SeCl<sub>6</sub>

$$\begin{array}{rcl} {\rm SeCl}_4 \ + \ 2 \ {\rm NH}_4 {\rm Cl} \ = \ ({\rm NH}_4)_2 {\rm SeCl}_6 \\ {\rm 220.78} & {\rm 106.99} & {\rm 327.77} \end{array}$$

For example, 5 ml. of an aqueous solution of 0.55 g. of  $NH_4Cl$  (~10 mmoles) is added to 5 ml. of the SeCl<sub>4</sub> solution containing 5 mmoles of SeCl<sub>4</sub>, and HCl gas is passed through at 0°C with frequent shaking. The [SeCl<sub>8</sub>]<sup>2-</sup> concentration increases with increasing HCl concentration and determines the point of precipitation. Precipitation of the deep yellow complex salt, which soon commences, is nearly quantitative if the solution is saturated with HCl. The solution is then colorless.

K<sub>2</sub>SeCl<sub>6</sub>

For example, 15 ml. of a concentrated hydrochloric acid solution of 0.75 g. of KCl (~10 mmoles) is added to 5 ml. of the SeCl<sub>4</sub> solution and the complex is precipitated as discussed under  $(NH_4)_2SeCl_5$ .

To isolate these moisture-sensitive complexes, the apparatus described for the preparation of hexachlorotitanium salts (compare section on Titanium) can be used.

 $Tl_2SeCl_6$  cannot be made by reaction in hydrochloric acid solution, since the Se<sup>4+</sup> is reduced to elemental Se.

 $Rb_2SeCl_{e}$  and  $Cs_2SeCl_{e}$  are analogously obtained from concentrated hydrochloric acid solutions even at room temperature.

PROPERTIES:

 $[(NH_4)_2SeCl_g]$  and  $K_2SeCl_g]$ : Yellow octahedra with an average diameter of 0.03 mm. Very sensitive to moisture; when moistened with hydrochloric acid, complete decomposition and loss of color occur in air in about 10 minutes. Very readily soluble in water with complete hydrolysis to selenic and hydrochloric acids; nearly

insoluble in concentrated hydrochloric acid. The complex salt reprecipitates on passage of HCl through a not too dilute aqueous solution at  $0^{\circ}$ C.

Crystal structure: K<sub>2</sub>PtCl<sub>6</sub> type.

**REFERENCES:** 

W. Petzold, Z. anorg. allg. Chem. 209, 267 (1932).

G. Engel, Z. Kristallogr. 90, 341 (1935).

H. Leibiger, Thesis, Freiburg i. Br., 1951.

## Diselenium Dibromide

## $Se_2Br_2$

I.

 $2 \text{ Se} + \text{Br}_2 = \text{Se}_2 \text{Br}_2$ 157.92 159.83 317.75

A round-bottom flask, equipped with a separatory funnel and gas outlet tube connected to a  $P_2O_5$  drying tube, is filled with a suspension of 20 g. of pure powdered Se in 50 ml. of dry CS<sub>2</sub>; 20 g. of pure Br<sub>2</sub> is then slowly added from the separatory funnel. If the flask is occasionally shaken, the reaction is soon complete. A reddish-brown solution is formed, from which the CS<sub>2</sub> is evaporated in vacuum as rapidly as possible. The product is deep-red, pure Se<sub>2</sub>Br<sub>2</sub>.

II.

 $3 \text{ Se} + \text{SeO}_2 + 4 \text{ HBr} = 2 \text{ Se}_2 \text{Br}_2 + 2 \text{ H}_2 \text{O}$ 236.88 110.96 323.70 635.50

A wide-neck flask is filled with 400 ml. of  $H_2O$  and this is saturated at 0°C with  $SO_2$ . While the  $SO_2$  passage is continued, 135 ml. of  $Br_2$  is gradually added from a separatory funnel. When the  $Br_2$  is completely consumed and the solution becomes colorless, 115 g. of  $SeO_2$  and then 235 g. of Se are added at room temperature. About one liter of concentrated  $H_2SO_4$  is then added, the reaction mixture is allowed to cool, and the dark red, oily layer is removed in a separatory funnel. The product is purified by repeated washing with concentrated  $H_2SO_4$ . The yield of  $Se_2Br_2$ is about 90%.

SYNONYMS:

Older names are selenium monobromide and selenous bromide.

**PROPERTIES:** 

Dark red, almost black, oily liquid with an unpleasant odor; hygroscopic; rapidly liberates Br<sub>2</sub> in air, simultaneously separating

426

Se. Partially decomposed on heating; first, some  $Br_2$  escapes, then some  $SeBr_4$  sublimes, and between 225 and 230°C a part of the  $Se_2Br_2$  boils without decomposition, leaving a residue of Se. d (15°C) 3.604.

In water,  $Se_2Br_2$  sinks to the bottom in oily drops and gradually decomposes into Se,  $SeO_2$  and HBr; soluble in  $CS_2$  and  $CHCl_3$ .

**REFERENCES:** 

- J. Meyer and V. Wurm, Z. anorg. allg. Chem. <u>190</u>, 90 (1930);
   R. Schneider, Pogg. Ann. 128, 327 (1866).
- II. V. Lenher and C. H. Kao, J. Amer. Chem. Soc. 47, 772 (1925).

#### Selenium Tetrabromide

#### SeBr<sub>4</sub>

 $Se + 2 Br_2 = SeBr_4$ 78.96 319.66 398.62

The reaction is carried out in a round-bottom flask equipped with a separatory funnel and gas outlet tube connected to a  $P_2O_5$ glass wool drying tube. Pure  $Br_2$  (100 g.) is allowed to flow slowly into a suspension of 20 g. of pure powdered Se in 50 ml. of dry  $CS_2$ . Finely crystalline yellow  $SeBr_4$  separates. When the addition is complete, a gas inlet tube is substituted for the separatory funnel and the  $CS_2$  and excess  $Br_2$  are driven off with a dry air stream. The residual  $SeBr_4$  is rapidly transferred into a tightly sealed vessel.

#### **PROPERTIES:**

Fine, ochre-yellow crystals with an unpleasant odor reminiscent of  $S_2Cl_2$ ; hygroscopic; decomposes in moistair to  $Br_2$ ,  $Se_2Br_2$  and reddish-brown Se. On heating,  $Br_2$  evolves. Between 75°C and 80°C, a mixture of SeBr<sub>4</sub> and Se<sub>2</sub>Br<sub>2</sub> sublimes, forming black, shiny crystals.

With water, SeBr<sub>4</sub> forms a clear solution of  $H_2$ SeO<sub>3</sub> and HBr.

**REFERENCES:** 

J. Meyer and V. Wurm, Z. anorg. allg. Chem. <u>190</u>, 90 (1930). See also R. Schneider, Pogg. Ann. 129, 450 (1866).

## Selenium Dioxide

#### SeO<sub>2</sub>

I.

$$\begin{array}{c} Se &+ O_2^{(NO_2)} \\ Se032.00 & 110.96 \end{array}$$

A large porcelain boat containing 60 to 75 g. of Se is pushed into the front end of a Pyrex glass tube (50-70 cm. long, 4-5 cm. in diameter); the tube is closed with a one-hole rubber stopper, and a fast O<sub>2</sub> stream, which has passed successively through a wash bottle containing fuming HNO, and a drying tube containing P<sub>2</sub>O<sub>z</sub>-glass wool, is introduced. A two-liter, two-neck glass bulb is mounted over the rear end of the tube. The front end of the boat is heated with a fishtail burner so that the Se melts and, shortly afterward, ignites, forming a long, luminous blue flame. The SeO<sub>2</sub> deposits to the rear of the tube, behind the boat. The O<sub>2</sub> stream must be very fast or elemental Se will also distill off and darken the SeO<sub>2</sub>. To avoid plugging of the tube, the sublimate is driven forward from time to time by means of a second burner. Traces of SeO<sub>2</sub>, which are carried along in the gas stream, are retained in the glass bulb. When all the Se has burned (about 1.25 hours) and only impurities remain in the boat, the product is repeatedly sublimed in the same tube in a pure O<sub>2</sub> stream until it is snow white; this removes small amounts of Se and nitrogen oxide impurities. The SeO<sub>2</sub> is then very pure and completely dry. The yield is about 90%, based on 100% pure Se.

Pure SeO<sub>2</sub> can be stored without decomposition only in carefully cleaned, well-sealed glass containers; even dust reduces it partially to Se. Very hygroscopic.

II.

 $Se + 4 HNO_3 = H_2SeO_3 + 4 NO_2 + H_2O$ 78.96 252.06 128.98

$$H_2SeO_3 = SeO_2 + H_2O_{128.98} + 110.96_{18.02}$$

Pure Se is slowly added to pure, concentrated, heated nitric acid, the solution is evaporated to dryness on a sand bath, and the residue is heated until sublimation begins. It is then absorbed in water and, to remove all  $H_2$  SeO<sub>4</sub> present, Ba(OH)<sub>2</sub> solution is added dropwise until the precipitate ceases to form. The solution is filtered and reevaporated to dryness while stirring. The crude product is powdered and repeatedly sublimed. For very pure material the compound is sublimed in pure O<sub>2</sub> as in method I. For many purposes, however, it is sufficient to sublime it two or three times from a porcelain dish into an inverted beaker or an inverted funnel containing a glass wool plug in the stem. The SeO<sub>2</sub> obtained in this manner, in contrast to that obtained in the dry oxidation process I, is not completely anhydrous, but the moisture can be removed by subsequent heating for 3-4 hours in an air stream at  $150^{\circ}$  C.

PROPERTIES:

White, crystalline substance; poisonous; hygroscopic, forms  $H_2SeO_3$  with moisture; readily discolors in the presence of small amounts of reducing materials (e.g., dust), separating free Se. Sublimes at 315°C, forming shiny needles. Can be melted without decomposition in a closed tube at somewhat higher temperature. d 3.95.

Readily soluble in water (at  $20^{\circ}$ C about 70 wt.% SeO<sub>2</sub>) and alcohol.

Crystallizes in C 47 structure type.

**REFERENCES:** 

- I. J. Meyer, Ber. dtsch. chem. Ges. 55, 2082 (1922).
- II. J. Thomsen, Ber. dtsch. chem. Ges. 2, 598 (1869); V. Lenher, J. Amer. Chem. Soc. 20, 555 (1898); J. Jannek and J. Meyer, Z. anorg. allg. Chem. 83, 51 (1913); L. M. Dennis and J. P. Koller, J. Amer. Chem Soc. 41, 949 (1919); O. Hönigschmid and W. Kapfenberger, Z. anorg. allg. Chem. 212, 198 (1933).

## Selenium Oxychloride

 $SeOCl_2$ 

 $SeO_2 + 2 HCl = SeO_2 \cdot 2 HCl$ 110.96 72.93 183.89

Carefully dried HCl is introduced into a weighed, 150-ml. round-bottom flask containing 50 g. of SeO<sub>2</sub> until all the SeO<sub>2</sub> dissolves, forming SeO<sub>2</sub> • 2HCl (theoretical weight increase, 33 g.). The flask must be occasionally shaken and protected from moisture. Since this reaction may be violent, it is best to insert an empty 500-ml. flask between the reaction flask and the wash bottles at the inlet (which are filled with concentrated H<sub>2</sub>SO<sub>4</sub> in order to dry the HCl gas). To dehydrate the SeO<sub>2</sub> • 2HCl, it is treated in the reaction flask for 10 minutes with 10 ml. of concentrated H<sub>2</sub>SO<sub>4</sub> while being slowly warmed to  $50^{\circ}$ C and constantly shaken. The partially dehydrated SeO<sub>2</sub> • 2HCl settles as the lower layer; it is then separated in a 100-ml. separatory funnel. The dehydration of the separated layer is repeated under the same conditions, each time with 5 ml. of concentrated  $H_2SO_4$ , until no further separation takes place. It should be noted that too large an excess of  $H_2SO_4$ , as well as insufficient dehydration, results in low yields and difficulties in the subsequent distillation.

Before the distillation, the cherry-red product is chlorinated until its color becomes straw yellow. In this process, the  $SeCl_2$ is converted to  $SeCl_4$ , which in turn reacts with  $SeO_2$  to form  $SeOCl_2$ . The chlorinated product is distilled twice in aspirator vacuum, using a boiling capillary. The distillation must be protected from moisture; the first milliliter is always discarded. The distillation is continued until fairly large quantities of  $SeO_2$ accumulate, causing heavy bumping. If the product ceases to be straw yellow before the second distillation, additional chlorine may have to be introduced.

Caution: All safety rules must be observed during distillation (safety goggles!) since violent decomposition may occur if the SeOCl<sub>2</sub> was incompletely dehydrated. For this reason, it is always necessary to determine, immediately prior to the distillation, whether the SeOCl<sub>2</sub> is completely dehydrated. (When heated in concentrated  $H_2SO_4$ , SeOCl<sub>2</sub> must form a clear solution.)

The yield is about 50 g., i.e., 68% based on SeO<sub>2</sub> input.

Since the pure product is very hygroscopic, it is transferred in the absence of moisture and stored in glass ampoules sealed off with a torch.

SYNONYM:

Selenyl chloride.

PROPERTIES:

Straw yellow liquid which fumes in moist air; hygroscopic; corrosive to skin. M.p. 11°C, b.p. 179°C with slight decomposition. d (20°C) 2.43. Hydrolyzes in water to form  $H_2SeO_3$  and HCl; completely miscible with CCl<sub>4</sub>, CHCl<sub>3</sub>, CS<sub>2</sub>, benezene and toluene.

Because of its dissolving ability, SeOCl<sub>2</sub> is occasionally used as a solvent for many substances.

**REFERENCE:** 

G. B. L. Smith and J. Jackson in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-London, 1950, p. 130.

## Selenous Acid (anhydrous)

H<sub>2</sub>SeO<sub>3</sub>

 $SeO_2 + H_2O = H_2SeO_3$ 110.96 18.02 128.98

Pure  $SeO_2$  (p. 428) is placed in a porcelain dish and dissolved in a small amount of water. The solution is concentrated on a water bath with careful exclusion of dust particles (danger of reduction to Se) until crystallization starts. After cooling, the separated  $H_2SeO_3$  is filtered by suction through a fritted glass filter and recrystallized from water. The pure product is pressed between filter papers and dried for several days in a vacuum desiccator over KOH. On prolonged standing over concentrated  $H_2SO_4$  or  $P_2O_5$ , further dehydration to  $SeO_2$  occurs.

PROPERTIES:

Colorless, prismatic crystals, which lose water in dry air, forming SeO<sub>2</sub>, but gradually liquefy in the presence of moisture; poisonous; easily reduced (even by dust) to Se. Melts at about  $70^{\circ}$ C to form a light yellow solution of SeO<sub>2</sub> in H<sub>2</sub>O. d  $^{15}_{4}$  3.00. Very readily soluble in water.

Crystal form: hexagonal.

**REFERENCES:** 

- A. Rosenheim and L. Krause, Z. anorg. allg. Chem. <u>118</u>, 177 (1921).
- J. Jannek and J. Meyer, Z. anorg. allg. Chem. 83, 51 (1918).

## Sodium Selenite

#### Na<sub>2</sub>SeO<sub>3</sub> · 5 H<sub>2</sub>O

 $\begin{array}{rll} {\rm H_2SeO_3} \ + \ 2 \ {\rm NaOH} \ = \ {\rm Na_2SeO_3} \ + \ 2 \ {\rm H_2O} \\ & (5 \ {\rm H_2O}) \\ {\rm 128.98} & 80.01 & 263.04 \end{array}$ 

A concentrated aqueous solution of selenous acid is mixed with the stoichiometric quantity of carbonate-free sodium hydroxide solution. The mixture is concentrated at room temperature in vacuum over CaCl<sub>2</sub>, and crystallization is induced by occasionally rubbing the vessel wall with a glass rod. The salt tends to form supersaturated solutions and then precipitates from these in microscopically small needles, which agglomerate into bundles. If these are used for seeding a saturated solution and the latter is allowed to evaporate further, transparent prisms up to 3 mm. long may be obtained. The crystals are filtered by suction through a fritted glass filter and dried on clay in a desiccator (do not place drying agents in the desiccator). The Na<sub>2</sub>SeO<sub>3</sub> · 5H<sub>2</sub>O thus obtained is analytically pure. PROPERTIES:

White crystals, needle-shaped to prismatic; stable in moist air; in dry air, loses water at the surface; poisonous; sensitive to reducing agents. On heating to 40°C, converts to the anhydrous salt.

Very readily soluble in water; 100 g. of solution contains about 68 g. of  $Na_2SeO_3 \cdot 5H_2O$  at 20°C.

**REFERENCE:** 

J. Janitzki, Z. anorg. allg. Chem. 205, 49 (1932).

#### Selenic Acid

#### H<sub>2</sub>SeO<sub>4</sub>

 $\begin{array}{rrrr} SeO_2 \ + \ H_2O_2 \ = \ H_2SeO_4 \\ 110.96 & 34.02 & 144.98 \end{array}$ 

The oxidation is carried out in aqueous solution so that dilute selenic acid solution is obtained first; this may be concentrated to the anhydrous acid by evaporation.

A solution of 150 g. of pure SeO<sub>2</sub> (p. 428) in 100 ml. of distilled water is prepared in a one-liter ground glass flask with a side gas inlet tube extending to the bottom. Then 500 g. of 30% H<sub>2</sub>O<sub>2</sub> (Perhydrol) is slowly introduced. The mixture is refluxed for 12 hours while O<sub>2</sub> is passed through. The selenic acid solution formed contains traces of H<sub>2</sub>SeO<sub>3</sub> as the only impurity.

To concentrate the solution, most of the water is distilled off in a slow stream of  $P_2O_5$ -dried air and aspirator vacuum until the temperature of the solution reaches 150°C (about four hours are quired for this). The acid concentration is then 85-90%. For further dehydration it is distilled at 1-2 mm.; the temperature in the flask should not exceed 160°C, or decomposition to H<sub>2</sub>SeO<sub>3</sub> occurs. To avoid local overheating at the flask walls above the liquid level, it is best to heat on a small hotplate and not in an oil bath. When no further H<sub>2</sub>O flows into the condenser and the receiver, the oily liquid is transferred to a flat dish and seeded at 10-15°C with a small amount of solid  $H_2SeO_4$ . The latter is obtained by cooling a few milliliters of the solution in a Dry Ice bath. To crystallize the solution, it is allowed to stand in a desiccator over  $P_2O_5$ . The more complete the dehydration during distillation, the more readily will the acid solidify. The crystals are completely dehydrated by standing in a  $P_2O_5$ -dried air stream at 5 mm. for several days. The yield is about 190 g. of 97-98% H<sub>2</sub>SeO<sub>4</sub> containing about 2-3% H<sub>2</sub>SeO<sub>3</sub>.

**PROPERTIES:** 

Anhydrous H<sub>2</sub>SeO<sub>4</sub> forms colorless prismatic or needle-shaped crystals; extremely hygroscopic. M.p. 58°C; the melt tends to supercool. On heating, decomposes into SeO<sub>2</sub>, O<sub>2</sub> and H<sub>2</sub>O. d (solid) (15°C) 2.95, d (liq.) (15°C, supercooled) 2.60. Readily soluble in water.

REFERENCES:

- L. I. Gilbertson and G. B. King in: L. F. Audrieth, Inorg. Syntheses, Vol. III, p. 137, New York-London 1950.
- L. I. Gilbertson and G. B. King, J. Amer. Chem. Soc. <u>58</u>, 180 (1936).

## Sodium Selenate

Na<sub>2</sub>SeO<sub>4</sub>

 $\begin{array}{rrrr} H_2 SeO_4 &+ & Na_2 CO_3 &= & Na_2 SeO_4 &+ & CO_2 &+ & H_2 O\\ 144.98 & & 106.00 & & 188.95 \end{array}$ 

About 210 g. of 85-90% selenic acid solution (p. 432) is used to neutralize 125 g. of C.P.  $Na_2CO_3$ . After filtering, the salt solution is evaporated (with heating) until a significant quantity of sediment is formed. If the solution acquires a slight reddish or brownish color due to colloidal Se, it is rediluted with  $H_2O$ , boiled and filtered after standing for 1-2 days. It must then be reconcentrated. The  $Na_2SeO_4$  is suction-filtered from the mother liquor, which is kept as warm as possible (the temperature of the solution must not fall below 45° C), and dried by pressing between filter papers. The salt is analytically pure.

At lower temperature (cooling of the salt solution in ice and filtering while cold), the decahydrate is obtained.

PROPERTIES:

White crystals, stable in air; isomorphous with  $Na_2SO_4$ . d 3.21. Very readily soluble in water.

**REFERENCE:** 

J. Meyer and W. Aulich, Z. anorg. allg. Chem. 172, 321 (1928).

# Sodium Selenopentathionate

Na<sub>2</sub>SeS<sub>4</sub>O<sub>6</sub> · 3 H<sub>2</sub>O

 $= \underbrace{ \underset{349.22}{Na_2SeS_4O_6} + \underset{270.26}{Na_2S_4O_6} + 4 CH_3COONa}_{328.16} + 2 H_2O}_{328.16}$ 

A solution of 130 g. of  $Na_2S_2O_3 \cdot 5H_2O$  in 40 ml. of  $H_2O$  is prepared with heating; the solution is cooled to 25-30°C and is added dropwise over a period of 20 minutes with mechanical stirring to an ice-salt cooled solution of 17.2 g. of FeO<sub>2</sub> and 20 ml. of H<sub>2</sub>O in 100 ml. of glacial acetic acid. The reaction temperature may not exceed  $0^{\circ}$ C. It is of utmost importance that the addition of Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> slow since thiosulfate catalyzes the decomposition of the be selenopentathionate unless there is a constant excess of H<sub>2</sub>SeO<sub>2</sub>. The clear, viscous, yellow-green solution is then mixed with 150 ml. of ethanol; after crystallization begins, 50 ml. of ether is added and the mixture is stirred and cooled for an additional 15 minutes. The crude product, containing about 40 g. of  $Na_2SeS_4O_6$ .  $3H_2O$  and 4 mole percent of  $Na_2S_4O_6$ , is filtered off, washed with ethanol and ether, and dried in vacuum over H2SO4. To recrystallize the salt, it is dissolved in 50 ml. of 0.2N HCl at  $30^{\circ}C$  and suction-filtered; 100 ml. of methanol is added to the filtrate and the mixture is cooled in an ice-salt bath. Pure Na<sub>2</sub>SeS<sub>4</sub>O<sub>6</sub> • 3H<sub>2</sub>O crystallizes; it is filtered, washed with ethanol, and dried in vacuum over H<sub>2</sub>SO<sub>4</sub>.

The yield is 25 g. (55%).

#### **PROPERTIES:**

Small, pale yellowish-green flakes. Very readily soluble in water, very soluble in methanol, insoluble in ethanol. Aqueous solutions are yellow-green and decompose gradually into selenium and tetrathionate; they can be stabilized by mineral acids; alkalis accelerate the decomposition. The pure salt can be stored over  $H_2SO_4$  for several months without decomposition (Se sometimes separates); readily loses its water of crystallization in vacuum over  $H_2SO_4$ .

#### **REFERENCE:**

O. Foss in: H. S. Booth, Inorg. Syntheses, Vol. IV, p. 88, New York-London-Toronto, 1953.

## Selenium Sulfur Trioxide SeSO3

 $Se + SO_3 = SeSO_3$ 78.96 80.07 159.03

Portions of liquid SO<sub>3</sub> are mixed with small quantities of selenium in a glass tube fused shut at one end and externally cooled with tap water. Between additions, the tube opening is closed with a one-hole stopper leading to a  $P_2O_5$ -filled absorption flask. The reaction is exothermic; the temperature must be so adjusted that the SO<sub>3</sub> is just above its solidification point (about 15°C). The SeSO<sub>3</sub> product settles to the bottom and on the walls as a thick, dark-green oil which eventually solidifies. When all the Se has reacted, the supernatant clear SO<sub>3</sub> is decanted and any traces are removed by suction-filtration. The remaining scaly SeSO<sub>3</sub> is rapidly loosened from the walls with a sharp-edged glass rod and transferred to ampoules, which are sealed.

PROPERTIES:

Dark-green crystalline substance; stable for some time at room temperature without decomposition; sensitive to moisture. On heating,  $SeSO_3$  decomposes, becoming first brown then yellow, then orange and finally red, and forming  $SO_2$ ,  $SeO_2$  and Se.

Violent reaction with water, yielding  $H_2SO_4$ ,  $H_2SO_3$ ,  $H_2SeO_3$ and Se. Soluble in oleum and concentrated sulfuric acid, giving a green solution.

**REFERENCES:** 

R. Weber, Pogg. Ann. 156, 531 (1875).

E. Divers and M. Shimosé, J. Chem. Soc. (London) <u>45</u>, 201 (1884); Ber. dtsch. chem. Ges. 17, 858 (1884).

## Selenium Nitride

 $Se_4N_4$ 

*Caution!* Se<sub>4</sub>N<sub>4</sub> is a very explosive substance. Even very small mechanical disturbances and mild action of chemicals cause extremely violent explosive decomposition. The strictest safety precautions must therefore be observed in handling this material.

REACTION OF AMMONIA WITH DIETHYL SELENITE IN BENZENE SOLUTION

The starting materials for preparing diethyl selenite are  $C_2H_5ONa$  and  $SeOCl_2$ . A solution of 20 g. of Na in 200 ml. of

absolute  $C_2 H_5 OH$  is prepared in a reflux apparatus. When the reaction is complete, most of the  $C_2 H_5 OH$  is distilled off and 70 g. of pure SeOCl<sub>2</sub> (p. 429) is allowed to drip onto the slurry; the latter is cooled with an ice bath and is frequently swirled around. The product, which has a strawberry color because of a slight Se precipitate, is extracted with ether and the ether solution dried with Na<sub>2</sub>SO<sub>4</sub>. Most of the ether is distilled off on a water bath, and the residue is fractionated twice in aspirator vacuum. The ester, a water-clear liquid, is collected between 83 and 85°C at 14 mm.

The conversion to  $Se_AN_A$  is carried out in a 100-ml., roundbottom flask, closed with a three-hole rubber stopper. The stopper holds a separatory funnel, a gas inlet tube extending to the bottom, and a gas outlet tube leading to a drying tower filled with soda lime. Sodium-dried benzene (20 g.) is placed in the flask, and the apparatus is flushed with dry NH<sub>3</sub> until all air is displaced. The NH<sub>3</sub> stream is continued and 3.1 g. of diethyl selenite is slowly added drop-by-drop. An initially green suspension forms. The color soon turns to red-brown as N<sub>2</sub> is evolved. After 1.75 hours, the precipitate is suction-filtered and carefully dried at 105°C. To remove SeO<sub>2</sub> and Se, it is then successively washed with water, 10% KCN solution, and finally again with water to remove CN. Pure Se<sub>4</sub>N<sub>4</sub> is best stored under benzene because of its explosive nature. The dry substance may not be placed in glass stoppered bottles since the contents generally explode when such bottles are opened. Cardboard boxes are the best containers.

Other preparative methods: Reaction of  $SeO_2$ ,  $SeCl_4$  or  $SeBr_4$  with liquid ammonia in a steel autoclave at 70-80°C (Jander and Doetsch).

#### PROPERTIES:

Formula weight 371.87. Orange-red, amorphous powder; becomes crystalline after prolonged standing under benzene; under the influence of light pressure or strong heating, explodes with great brisance.

Insoluble in water; slowly decomposed by boiling water to yield  $H_2SeO_3$ , Se and  $NH_3$ . Slightly soluble in glacial acetic acid.

REFERENCES:

- W. Strecker and H. E. Schwarzkopf, Z. anorg. allg. Chem. <u>221</u>, 193 (1934).
- H. E. Schwarzkopf, Thesis, Marburg, 1932.
- J. Jander and V. Doetsch, Angew. Chem. 70, 704 (1958).
- J. Jander and V. Doetsch, Chem. Ber. (in press).

## Tellurium

#### Te

#### VERY PURE TELLURIUM

Commercial tellurium must generally be further purified for most laboratory uses since it contains some  $\text{TeO}_2$  as well as small amounts of Se, S and heavy metals (particularly Cu, Pb and Ag). It should be noted that Te is a strong respiratory poison. The metal is distilled in a hydrogen stream and converted to the readily crystallizable basic salt  $\text{Te}_2\text{O}_3(\text{OH})\text{NO}_3$ ; the latter is purified by repeated recrystallization, ignited to the oxide, and reduced in hydrochloric acid solution with hydrazine to form elemental Te. To achieve an ultrapure product, the metal is then distilled in high vacuum.

Prior to the prepurification by distillation in a hydrogen stream, the Te is pulverized in an agate mortar and placed in a large quartz boat inserted in the forward section of a quartz tube. The boat is gradually heated to red heat while a moderately fast stream of pure,dry hydrogen is passed over it. The Te melts and, with increasing temperature, gives off a greenish-yellow vapor which is carried along by the  $H_2$  and condenses in small metallic balls in the colder section of the tube. The distillation is interrupted when about 90% of the Te has vaporized. All less volatile metals are found in the residue. After cooling in a stream of  $H_2$ the distilled Te can be readily loosened from the tube wall and taken out with platinum forceps.

The prepurified product is dissolved in concentrated hydrochloric acid containing some nitric acid; excess nitric acid is decomposed by prolonged heating and the solution is diluted with water, taking care not to exceed the point at which hydrolysis (TeO<sub>2</sub> separation) would occur. The solution is filtered to remove any impurities which may be present, and is then reduced with freshly distilled hydrazinium hydroxide solution. The Te precipitate is washed with water and then alcohol and dried in a vacuum desiccator over concentrated  $H_2SO_4$ . The finely divided metal is then dissolved in nitric acid (d 1.25) at 70°C. Higher temperatures are to be avoided because of the precipitation of considerable quantities of rather insoluble TeO<sub>2</sub>. As the solution is concentrated, the basic nitrate Te<sub>2</sub>O<sub>3</sub>(OH)NO<sub>3</sub> precipitates in large, well formed crystals. The salt is again recrystallized from nitric acid of the same concentration. After drying, it is ignited to TeO<sub>2</sub> in a procelain crucible in an electric furnace. The dioxide is dissolved in hydrochloric acid (d 1.12) and reduced to Te with hydrazinium hydroxide solution; the Te is washed and dried as above. Since the metal powder is always partially reoxidized

to  $\text{TeO}_2$  by atmospheric oxygen, the product powder is melted in a quartz boat while a pure hydrogen stream is passed over it. It is kept liquid until all the  $\text{TeO}_2$  is reduced and the whole surface is shiny. To obtain an ultrapure product, the metal can then be distilled from a quartz boat placed in a quartz tube closed at one end; this is done in high vacuum and at as low a temperature as possible.

The purity of the product is determined most reliably by spectral analysis.

A process for preparing Te single crystals is given by Schmid and Wassermann; it involves melting the pure metal in narrow tubes, followed by very slow solidification.

**REFERENCES:** 

- O. Hönigschmid, R. Sachtleben and K. Wintersberger, Z. anorg. allg. Chem. <u>212</u>, 242 (1933).
- O. Hönigschmid and H. Baudrexler, Z. anorg. allg. Chem. 223, 91 (1935).

A. Stähler and B. Tesch, Z. anorg. allg. Chem. 98, 1 (1916).

E. Schmid and G. Wassermann, Z. Phys. <u>46</u>, 653 (1928).

## **Colloidal Tellurium Solution**

Stable Te sols are obtained by the reduction of telluric acid with hydrazinium hydroxide.

A solution of 2-3 g. of very pure  $H_6 TeO_8$  (p. 451) in one liter of very pure water (see section on Hydrogen, Deuterium, Water, p. 117) is prepared and heated on a water bath to 40-50°C. Higher temperatures may cause a yellow color in the subsequent reduction. A very dilute aqueous  $N_2H_4$  solution (1 : 2000) is added dropwise to the warm  $H_8TeO_6$  solution until the color of the hydrosol no longer changes. An excess of reducing agent should be avoided since it renders the sol very unstable and causes coagulation. The liquid is transferred into a dialyzer or parchment paper bag and is dialyzed until completely pure; the exterior water is frequently renewed.

**REFERENCES:** 

A. Gutbier, Z. anorg. allg. Chem. <u>32</u>, 51 (1902); Kolloid-Z. <u>4</u>, 180 (1909).

## Hydrogen Telluride

## H<sub>2</sub>Te

Since  $H_2Te$  is a poisonous gas with an unpleasant odor and, when inhaled in large quantities, greatly irritates the bronchial



Fig. 169. Preparation of hydrogen telluride.

tubes and damages the nervous system, its preparation must be carried out in a good hood and in carefully sealed apparatus.

## I. CATHODIC REDUCTION OF Te

Electrolytic preparation of  $H_2$ Te is generally preferred to the method of acid decomposition of tellurides (see II) because the yield is higher.

The apparatus of Fig. 169 is used. The electrolysis flask proper, which is made of glass, stands in a zinc tub surrounded by thermal insulation and an external wooden box. The cathode is introduced from below through adapter tube C. The cathode is made from a thin-wall glass tube which is fused and closed at one end. A few grams of pure Te are melted in the tube and a copper wire is inserted before the melt solidifies. After cooling, the space above the Te is filled with plaster of Paris. The tube tip at the tellurium end is then cracked by scratching, warming it slightly, and immersing it in water. The tellurium is thus exposed. The anode A is of platinum foil. The electrolysis vessel is filled to the upper edge of the anode with 50% sulfuric acid and is cooled from the outside by a Dry Ice bath. The run is conducted in a darkened room since, according to Hempel and Weber, H<sub>2</sub>Te decomposes more rapidly in light. The electrolysis proceeds at 4.5 amp, and 75-110 v., and the pressure is so adjusted that only a thin layer of acid is found above the Te at the cathode. The electrolyte temperature is maintained at 0°C by balancing the exterior cooling and the heat produced by the current. The evolving gas mixture, which contains up to 45% H<sub>2</sub>Te (besides the H<sub>2</sub>), is dried by passage through two U tubes containing CaCl<sub>2</sub> and  $P_2O_5$ -glass wool (no rubber connections may be used), and is then condensed in a trap cooled with liquid nitrogen. As usual, the trap is protected against atmospheric moisture by a drying tube. The gas obtained after reevaporation is sufficiently pure for most purposes.

If a very pure product is required, the traces of inert gas are removed by repeated melting and solidification in vacuum by fractionation or sublimation in high vacuum; a considerable amount of the first and last cuts is rejected.

Hydrogen telluride is stored in the dark either as a solid at low temperature or in the vapor state in torch-sealed glass flasks. Mercury may not be used as a sealing liquid since it is attacked even by carefully dried  $H_2$ Te.

II.

 $Al_2Te_3 + 6 HCl = 3 H_2Te + 2 AlCl_3$ 436.77 218.79 388.88 266.68

The  $Al_2Te_3$  used as starting material is prepared from the elements according to the method of Moser and Ertl (see section on Aluminum).

The acid decomposition is carried out in the apparatus shown in Fig. 170, which permits the introduction of the telluride into the acid in an inert gas stream. All parts of the apparatus must be carefully dried before assembly since  $H_2Te$  is decomposed by moisture, with Te precipitation. Hydrochloric acid (4N) is boiled



Fig. 170. Preparation of hydrogen telluride.

and then cooled in a  $N_2$  stream. The reaction flask E (100 ml.) is filled with this acid. The entire experiment is done in an atmosphere of pure  $N_2$ , which is introduced through the side arm U and the hollow piston rod S whose conical end fits into the ground glass joint H of adapter A. When the air has been displaced from the apparatus, coarsely powdered  $Al_2Te_3$  is rapidly introduced at T. By tilting or slight tapping of the apparatus, the powder is gradually transferred into adapter A; by slight downward motion of the piston rod, small portions are introduced into the acid. At the end, flask E is heated to a moderate temperature for a short time in order to complete the gas evolution. The  $H_2Te$ passes through a small water-cooled condenser which forms a side

440

arm of E and through two drying tubes filled with CaCl<sub>2</sub> and P<sub>2</sub>O<sub>5</sub>glass wool, and is frozen in a trap cooled with liquid nitrogen and protected against moisture by a drying tube. When pure starting materials are used, the product is generally sufficiently pure; it can be further fractionated via the procedure given in method I.

#### PROPERTIES:

Formula weight 129.63. Colorless gas, unpleasant odor reminiscent of  $AsH_3$ ; poisonous. Decomposes with Te separation with even traces of air or moisture, also with cork and rubber. Whenever possible, ground glass joints should therefore be used. The liquid is instantaneously decomposed by light, which produces discoloration; whether or not light accelerates the decomposition of the gas as well is not certain as the data existing in the literature are contradictory. According to Moser, pure dry  $H_2$ Te is stable even in ultraviolet light.

M.p.  $-49^{\circ}$ C, b.p.  $-2^{\circ}$ C. Begins to decompose into the elements slightly above room temperature. d (liq.) ( $-12^{\circ}$ C) 2.68; weight per liter 6.234 g.

Soluble in water with rapid decomposition; the saturated solution is about 0.1N.

**REFERENCES:** 

- I. W. Hempel and M. G. Weber, Z. anorg. allg. Chem. <u>77</u>, 48 (1912); see also A. Klemenc, Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], Leipzig, 1938, p. 186; L. M. Dennis and R. P. Anderson, J. Amer. Chem. Soc. <u>36</u>, 882 (1914).
- II. L. Moser and K. Ertl, Z. anorg. allg. Chem. 118, 269 (1921).

## Sodium Telluride, Potassium Telluride

#### Na<sub>2</sub>Te, K<sub>2</sub>Te

2 Na	+ Te =	$Na_2Te$	2 K	+ Te $=$	K₂Te
45.99	127.61	173.60	78.19	127.61	205.80

The synthesis is carried out in liquid  $NH_3$  with exclusion of air and moisture. For the apparatus and procedure, see  $K_2S$  (p. 360). The directions may be followed in all details; the compounds  $Na_2Te$ and  $K_2Te$  prepared accordingly are analytically pure.

PROPERTIES:

Na<sub>2</sub>Te: White crystalline powder; very hygroscopic; decomposes immediately in air, becoming dark; crystallizes in the C1

structure type. M.p.  $953^{\circ}C$ ; d 2.90. Soluble in water; on contact with air, the solution rapidly precipitates black Te powder.

 $K_2$ Te: Faintly yellow, crystalline substance; hygroscopic; decomposes instantaneously on contact with air with Te precipitation. Crystallizes in Cl structure type. d 2.52. Soluble in water; the solution precipitates Te in air.

**REFERENCE**:

W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. <u>241</u>, 281 (1939).

## Sodium Ditelluride

Na<sub>2</sub>Te<sub>2</sub>

 $Na_2Te + Te = Na_2Te_2$ 173.60 127.61 301.21

The stoichiometric quantity of Te and 2.5 g. of Na<sub>2</sub> Te are introduced in a rapid stream of pure N<sub>2</sub> into a Pyrex tube, closed at one end. The tube is evacuated with a high-vacuum pump, the open end is melted shut under vacuum, and the tube is heated in an electric furnace to  $500^{\circ}$  C until a homogeneous melt is formed (about 45 min.). After cooling, a gray-black, very hard mass of Na<sub>2</sub> Te<sub>2</sub> is obtained.

PROPERTIES:

Gray-black, metallic, shiny microcrystalline substance; decomposes in the presence of air and moisture.

**REFERENCES:** 

W. Klemm, H. Sodomann and P. Langmesser, Z. anorg. allg. Chem. <u>241</u>, 281 (1939).

W. Klemm and H. Sodomann, Z. anorg. allg. Chem. 225, 273 (1935).

## **Tellurium Tetrachloride**

TeCl₄

 $\begin{array}{rrrr} Te \ + \ 2 \ Cl_{2} \ = \ TeCl_{4} \\ 127.61 & 141.83 & 269.44 \end{array}$ 

The synthesis is carried out in the glass apparatus shown in Fig. 171; because of the extreme hygroscopicity of  $TeCl_4$ , all



Fig. 171. Preparation of tellurium tetrachloride.

connections are glass fused to glass. The number of sealable ampoules  $A_1, A_2$ , etc., can be varied depending on the number of individual samples desired. The apparatus is dried in a drying oven for 12 hours prior to the run. Then 50 g. of pure Te (powder or coarse fragments of Te sticks) is placed in the still warm flask K through H. Opening H is then closed, c is clamped shut, and the whole apparatus is heated by fanning with a flame to remove the last traces of water film; at the same time, the apparatus is evacuated through d. When the apparatus is cooled down, the connection to the vacuum line is broken and the apparatus is rinsed with dry, O<sub>2</sub>-free N<sub>2</sub> introduced at c. When all air is displaced, a slow stream of dry, O<sub>2</sub>-free Cl<sub>2</sub> (p. 272) is introduced and K is carefully heated at the same time with a Bunsen burner to initiate the reaction. The flame can be removed as soon as the reaction starts. The contents of the flask liquefy after some time. They turn black at first, then transparent and dark red, and finally amber yellow. Toward the end of the reaction, K is again slightly heated. When the product has become pure yellow, dry Cl<sub>2</sub>-HCl mixture is passed through for some time with slight heating, in order to decompose any oxychloride that might have formed. At the end, the product is distilled into the ampoules in a slow Cl<sub>2</sub> stream, the heat being supplied by fanning with a flame. The ampoules are then sealed. The yield is 95 g. of pure TeCl.

#### PROPERTIES:

Fine, white crystals; very hygroscopic; deliquesces in moist air with partial hydrolysis. M.p. 224°C, b.p. 390°C; d 3.01. The melt is yellow, the vapor orange-red.

Hydrolyzes with water to form HCl and TeO<sub>2</sub>. Soluble in absolute alcohol and toluene.
**REFERENCES:** 

- J. F. Suttle and C. R. F. Smith in: L. F. Audrieth, Inorg. Syntheses, Vol. III, p. 140, New York-London 1950.
- J. F. Suttle and R. P. Geckler, J. Chem. Ed. 23, 135 (1946).
- O. Hönigschmid and H. Baudrexler, Z. anorg. allg. Chem. 223, 91 (1935).
- A. Stähler and B. Tesch, Z. anorg. allg. Chem. 98, 1 (1916).

# Hexachlorotellurium Salts

The required stock solution of  $\text{TeCl}_4$  in hydrochloric acid is best prepared by treating Te powder with aqua regia, evaporating this solution to dryness, and taking up residue in as little concentrated hydrochloric acid as possible.

(NH<sub>4</sub>)<sub>2</sub>TeCl<sub>6</sub>

$$TeCl_4 + 2NH_4O = (NH_4)_2TeCl_6.$$
  
269.43 106.99 376.42

Concentrated aqueous  $NH_4Cl$  solution is added to the  $TeCl_4$  solution in hydrochloric in such a quantity that the mixture is just at the point of  $NH_4Cl$  precipitation. On prolonged standing in air, the yellow complex salt precipitates in beautiful, relatively large octahedra. The precipitation can be accelerated and completed by passage of HCl and cooling. The salt is considerably less moisture-sensitive than the analogous selenium compound. After suction-filtration and pressing between filter papers, it can therefore be dried in air and stored in a desiccator.

K<sub>2</sub>TeCl<sub>6</sub>

$$\begin{array}{c} \text{TeCl}_4 + 2 \text{ KCl} = \text{K}_2 \text{TeCl}_6 \\ \text{269.43} & 149.10 & 418.53 \end{array}$$

The TeCl<sub>4</sub> solution is mixed with aqueous KCl solution in a ratio not exceeding one mole of KCl per mole of TeCl<sub>4</sub>; precipitation should not be allowed to occur. The precipitation procedure corresponds to that for  $(NH_4)_2 TeCl_6$ . Since the K salt is considerably more sensitive to atmospheric moisture than the NH<sub>4</sub> salt, it is best to dry it in the apparatus described for the preparation of hexachlorotitanium salts (see section on Titanium).

The Rb and Cs chloro complex salts are prepared analogously. For the preparation of  $Tl_2TeCl_6$ , see G. Engel.

PROPERTIES:

 $[(NH_4)_2 TeCl_8$  and  $K_2 TeCl_8]$ ; yellow, octahedral crystals, about 0.1 mm. in diameter; decompose gradually in air, particularly when moistened with hydrochloric acid, with color loss and HCl evolution. The NH<sub>4</sub> salt is considerably more stable than the K salt. Both are very readily soluble in water, hydrolyzing to tellurous and hydrochloric acids. Dissolve without decomposition in not too dilute hydrochloric acid, the K salt more readily than the NH<sub>4</sub> salt. The latter can be recrystallized from hydrochloric acid.

CRYSTAL STRUCTURE:

 $[(NH_4)_2 TeCl_6: K_2 PtCl_6]$  structure type;  $K_2 TeCl_6:$  optically a biaxially negative crystal; monoclinic, pseudocubic lattice; deviation from the  $K_2 PtCl_6$  structure type is only slight.

REFERENCES:

- W. Muthmann and J. Schäfer, Ber. dtsch. chem. Ges. <u>26</u>, 1004 (1893).
- H. L. Wheeler, Z. anorg. allg. Chem. 3, 434 (1893).
- A. Gutbier, F. Flury and H. Micheler, J. prakt. Chem. [2] <u>83</u>, 153 (1911).
- G. Engel, Z. Kristallogr. 90, 357 (1935).

## **Tellurium Tetrabromide**

## TeBr<sub>4</sub>

$$\begin{array}{rrrr} Te \ + \ 2 \ Br_2 \ = \ TeBr_4 \\ 127.61 \ 319.66 \ 447.27 \end{array}$$

Pure Te (5 g.) is introduced through adapter a into the elongated reaction flask A of the apparatus shown in Fig. 172. Adapter a is connected to a N<sub>2</sub> purification train which furnishes either pure dry N<sub>2</sub> or N<sub>2</sub> containing bromine vapor. For the latter, the gas stream may be passed through a wash bottle containing dry Br<sub>2</sub>, followed by a U tube containing  $P_2O_5$ -glass wool. First, the apparatus in purged with pure  $N_2$ . Then A is cooled with ice water and the N<sub>2</sub>-Br<sub>2</sub> mixture is introduced. A portion of the Br<sub>2</sub> condenses on the Te and reacts quietly with it, while the remainder is retained in the empty U tube D, cooled with an ice-salt mixture. The difference between the weight loss of the Br<sub>2</sub> wash bottle and the weight of the condensate in D indicates the amount of bromine remaining in A. When this becomes about twice the amount needed for quantitative conversion to TeBr<sub>4</sub>, the gas stream is interrupted, stopcock C is closed, and the product slurry is allowed to stand at room temperature for several hours (better overnight) in order to complete the reaction. Then C is reopened and the excess

bromine is purged by a stream of  $N_2$  while *A* is simultaneously heated to 50°C. To purify the product, it is sublimed in vacuum. Gas inlet tube *a* is sealed off at *d*, the whole apparatus is tilted to a horizontal position, and ground glass joint *E* is connected to an aspirator through a  $P_2O_5$  drying tube. During evacuation, *A* is heated to sublimation temperature (about 350°C) with an electric furnace. Any black condensate which may separate in *B* at 200°C is vaporized by heating with a burner. The subsequently deposited yellow to orange-red powder is quite pure TeBr<sub>4</sub>; if necessary, this can be further purified by an analogous sublimation in high vacuum. Because of its hygroscopicity, the product is either immediately sealed in the condensation receiver or is rapidly transferred to a well-sealed vessel.

PROPERTIES:

Yellow to orange, hygroscopic crystals. On heating, partially decomposes with  $Br_2$  evolution; for this reason,  $TeBr_4$  cannot be melted or distilled at atmospheric pressure without decomposition. d (15°C) 4.31.

Hydrolyzes in water; soluble in hydrobromic acid, ether and glacial acetic acid.

**REFERENCES:** 

- O. Hönigschmid, R. Sachtleben and K. Wintersberger, Z. anorg. allg. Chem. <u>212</u>, 242 (1933).
- B. Brauner, Mh. Chem. 10, 411 (1889).



Fig. 172. Preparation of tellurium tetrabromide.

## Tellurium Tetraiodide

TeI₄

 $\begin{array}{rrrr} Te(OH)_6 &+ & 6 & HI &= & TeI_4 &+ & I_2 &+ & 6 & H_2O \\ 229.66 & & 767.57 & & 635.29 & 253.84 \end{array}$ 

A very concentrated  $H_{\rm g}$ TeO<sub>g</sub> solution (p. 451) is mixed at room temperature with slightly more than the stoichiometric quantity of fuming hydriodic acid (d 2.00). A heavy, gray precipitate of TeI<sub>4</sub> immediately separates. It is suction-filtered on a fritted glass filter and freed of traces of hydriodic acid by pressing on clay. Concentration of the mother liquor at room temperature yields a considerable additional amount of the compound. When dry, the crystals are washed several times with pure CCl<sub>4</sub> to remove iodine and are finally pulverized under CCl<sub>4</sub> until the continuously renewed wash fluid no longer is colored. The product is then analytically pure.

PROPERTIES:

Iron-gray crystalline substance; stable even in moist air. Decomposes on heating, giving off  $I_2$ . M.p. (closed tube) 280°C; d (15°C) 5.05.

Hydrolyzed slowly in cold water, rapidly in warm water, forming TeO<sub>2</sub> and HI. Soluble in hydriodic acid, forming  $H(TeI_5)$ ; slightly soluble in acetone.

**REFERENCES:** 

A. Gutbier and F. Flury, Z. anorg. allg. Chem. <u>32</u>, 108 (1902). See also M. Damiens, Ann. Chim. [9] <u>19</u>, 44 (1923).

# **Tellurium Dioxide**

TeO<sub>2</sub>

I.

$$2 \text{ Te} + 9 \text{ HNO}_3 = \text{Te}_2\text{O}_3(\text{OH})\text{NO}_3 + 8 \text{ NO}_2 + 4 \text{ H}_2\text{O}$$
  
255.22 567.14

$$Te_{2}O_{3}(OH)NO_{3} = 2TeO_{2} + HNO_{3}$$

$$319.22$$

Concentrated nitric acid (95 ml.; d 1.42) is slowly added to a suspension of 20 g. of Te powder (commercial grade) in 200 ml. of distilled water in a one-liter beaker. The mixture is allowed to

stand for about 10 minutes with occasional shaking. Any impurities which have not dissolved after this time (selenides, tellurides, etc.) are rapidly suction-filtered, and the filtrate is mixed with another 65 ml. of concentrated nitric acid. The solution is boiled until all oxides of nitrogen are removed. If the Te contained any Sb or Bi, the basic nitrates of these elements separate and, if necessary, are filtered through a fritted glass suction filter. The clear solution is concentrated to 100 ml. on a water bath, using a 600-ml. beaker; it is then allowed to cool and is suction-filtered to remove the crystallized Te<sub>2</sub>O<sub>3</sub>(OH)NO<sub>3</sub>. The salt is washed with water and dried on clay in air. In order to convert it to TeO<sub>2</sub>, it is heated for two hours at 400-430°C (sand bath or hotplate) in a porcelain dish protected from dust by an inverted beaker (to avoid reduction to Te). The TeO<sub>2</sub> product is pure white and is analytically pure. If the starting material is very impure, the Te<sub>2</sub>O<sub>3</sub>(OH)NO<sub>3</sub> may again be recrystallized from nitric acid (d 1.25) before the ignition. Immediately after cooling, the pure product is transferred into a tightly sealed clean vessel to prevent any discoloration by the reducing action of organic substances in the atmosphere. Assuming that good quality commercial Te is used, the yield is about 21 g. or 84%.

II.

 $\begin{array}{rrrr} H_{6}TeO_{6} &=& TeO_{2} &+& 1/_{2}O_{2} &+& 3H_{2}O_{2}\\ 229.66 & 159.61 & 16.00 & 54.05 \end{array}$ 

Pure  $H_6 TeO_6$  (p. 451) is heated in a porcelain crucible. The heating is done in two stages: first the material is heated slowly to 150-200°C until most of the water is driven off; then it is ignited for some time at about 600°C. If the conversion to  $TeO_2$  is quantitative, the crucible contents are pure white after cooling. If the material is yellowish, the product still contains  $TeO_3$  and must be heated further. The purity of the  $TeO_2$  product is determined by that of the  $H_6 TeO_6$  used.

PROPERTIES:

White crystals; not hygroscopic; discolors in the presence of organic materials because of partial reduction. M.p. 733°C; the melt is dark yellow. d 6.02.

Very slightly soluble in water (about 1 : 150,000); soluble in concentrated mineral acids and alkalis, with salt formation.

Crystallizes in C 4 structure type.

**REFERENCES:** 

I. H. Marshall in: L. F. Audrieth, Inorg. Syntheses, Vol. III, p. 143, New York-London, 1950; see also P. L. Baynton, Nature <u>176</u>, 691 (1955). II. L. Staudenmaier, Z. anorg. allg. Chem. <u>10</u>, 189 (1895); W. Marckwald, Ber. dtsch. chem. Ges. 40, 4730 (1907).

## Tellurous Acid

## $H_2TeO_3$

 $\begin{array}{rl} K_{2}TeO_{3} \ + \ 2 \ HNO_{3} \ = \ H_{2}TeO_{3} \ + \ 2 \ KNO_{3} \\ 253.80 & 126.03 & 177.63 \end{array}$ 

The  $K_2 TeO_3$  stock solution is made by dissolving pure Te (p. 437) in dilute HNO<sub>3</sub>, evaporating to dryness, and dissolving the residue in 10% KOH. The solution is colored with one drop of phenolphthalein and mixed at 0°C with dilute HNO<sub>3</sub>, added dropwise from a burette, until it is colorless. A white — first flaky, then finely powdered — precipitate of  $H_2 TeO_3$  separates. It is filtered and thoroughly washed with ice water until the wash water is free of NO<sub>3</sub><sup>-</sup> and K<sup>+</sup> ions. The product is stored under distilled water since, in the dry state, it readily loses  $H_2O$ , forming TeO<sub>2</sub> hydrates of lower water contents.

**PROPERTIES:** 

White, amorphous substance of varying composition;  $H_2O$  content is frequently less than that corresponding to the formula  $H_2TeO_3$ ; loses water when dry but can be stored under  $H_2O$  for several days without change.

On heating above room temperature, extensively loses water with conversion to  $\text{TeO}_2$ . Very slightly soluble in water (at  $18^{\circ}\text{C}$  about  $3.0 \cdot 10^{-6}$  moles/liter).

REFERENCES:

J. Kasarnowsky, Z. phys. Chem. 109, 287 (1924).

E. B. R. Prideaux and J. O. N. Millott, J. Chem. Soc. (London) 1929, 2703.

#### Sodium Tellurite

Na<sub>2</sub>TeO<sub>3</sub>

$$TeO_2 + Na_2CO_3 = Na_2TeO_3 + CO_2$$
  
159.61 106.00 221.60 44.01

Stoichiometric quantities of pure  $TeO_2$  (p. 447) and C. P.  $Na_2CO_3$  are melted together in a Pt crucible. In order to prevent

oxidation to tellurate, the heating must be carried out in a  $CO_2$  atmosphere. As soon as gas evolution stops and a clear melt is obtained, the latter is allowed to cool under  $CO_2$ . The white crystalline mass is  $Na_2TeO_3$ .

PROPERTIES:

White crystalline substance; converts to  $Na_2TeO_4$  when heated in air.

Very readily soluble in water; the solution is decomposed by atmospheric  $CO_2$ , yielding  $TeO_2$ .

**REFERENCE:** 

V. Lenher and E. Wolesensky, J. Amer. Chem. Soc. 35, 718 (1913).

# Tellurium Trioxide

## TeO<sub>3</sub>

 $H_{6}TeO_{6} = TeO_{3} + 3H_{2}O_{229.66}$  175.61 54.05

I.  $\alpha$ -TeO<sub>3</sub>

Pure  $H_e TeO_e$  (p. 451) is slowly heated to 300-320°C in a porcelain crucible with occasional stirring. When all the  $H_2O$  has been driven off, the material is cooled and mixed several times with concentrated hydrochloric acid to remove any  $TeO_2$  which might have formed. The washing flask should be cooled. The pure  $TeO_3$ is then thoroughly washed with water and dried at 100°C. The yield is about 30%.

PROPERTIES:

Bright yellow powder which, in contrast to  $\beta$ -TeO<sub>3</sub>, gives no powder pattern.

On heating,  $\alpha$ -TeO<sub>3</sub> becomes brown at about 200°C, and above 400°C decomposes to TeO<sub>2</sub> and O<sub>2</sub>. d (15°C) 5.075.

Nearly insoluble in cold water, but noticeably soluble on long standing in hot water, forming  $H_6 TeO_6$ . Soluble in strong alkali, forming tellurates.

II.  $\beta$ -TeO<sub>3</sub>

Pure  $H_6 TeO_6$  is mixed with a few drops of concentrated  $H_2SO_4$ and heated for 12-15 hours at about 320°C in a torch-sealed tube. In order to remove  $\alpha$ -TeO<sub>3</sub>, the product is then boiled with concentrated KOH solution and the resulting solution is filtered with suction through a fritted glass filter. The  $\beta$ -TeO<sub>3</sub> product is washed thoroughly with water and dried at 100°C. The yield is about 40%.

PROPERTIES:

Gray, microcrystalline substance which, in general, is considerably less reactive than  $\alpha$ -TeO<sub>3</sub>. On heating above 400°C, decomposes into TeO<sub>2</sub> and O<sub>2</sub>. d 6.21.

Insoluble in water; even hot acids and concentrated alkali do not attack it and do not form salts.

**REFERENCES:** 

E. Montignie, Z. anorg. allg. Chem. <u>252</u>, 111(1943); <u>253</u>, 90 (1945). Bull. Soc. Chim. France, Mem. <u>1947</u>, 564.

## Telluric Acid

## H<sub>6</sub>TeO<sub>6</sub>

I.

The starting materials are very fine Te powder and aqueous HClO<sub>3</sub> solution. The acid required for the oxidation of 0.1 mole (that is, 12.75 g.) of Te is prepared by adding a lukewarm mixture of 40 ml. of H<sub>2</sub>O and 7.2 ml. of concentrated H<sub>2</sub>SO<sub>4</sub> to a solution of 24 g. of Ba(ClO<sub>3</sub>)<sub>2</sub>·H<sub>2</sub>O in 100 ml. of H<sub>2</sub>O. After about five hours, the solution is decanted from the precipitated BaSO<sub>4</sub> through a filter, and if desired the residue is extracted once with H<sub>2</sub>O. To oxidize the Te, it is added to a 500-ml., round-bottom flask placed under the hood. It is then moistened with 5 ml. of 50% HNO, and about one fourth of the HClO3 solution. With thorough agitation (by swirling) the flask contents soon boil and the reaction proceeds at boiling with strong Cl<sub>2</sub> evolution to form H<sub>6</sub>TeO<sub>6</sub>. The reaction should definitely not be slowed by intermittent cooling. If white flakes (H<sub>2</sub>TeO<sub>2</sub> or TeO<sub>2</sub>) should form in the liquid, the contents of the flask must be continuously boiled. As soon as the evolution of Cl<sub>2</sub> subsides, the rest of the HClO<sub>3</sub> solution is added in several large portions while the solution is again kept boiling and is constantly agitated. After 30 minutes, the reaction is complete and all the Te should be dissolved. The clear liquid is now concentrated in a porcelain dish, first over a free flame and

finally on a water bath, until crystals begin to separate (the solution volume is about 1/3 of the original at this point). The dish is then set on ice; the acid separates with stirring as a fine, pure-white crystalline powder. The precipitate is suction-filtered through fritted glass and dissolved once more in hot distilled  $H_2O$  in order to remove traces of HC1. The solution is mixed with a few drops of dilute AgNO<sub>3</sub>. After filtering off the AgCl, the solution is evaporated until crystallization begins. When the liquid is cooled slowly, the acid separates in beautiful, water-clear crystals up to 2 cm. long. It is suction-filtered through fritted glass, washed with water at 0°C, then with alcohol and ether, and dried in a vacuum desiccator over  $P_2O_5$ . By mixing the mother liquor with an equal volume of alcohol, a further finely crystalline fraction, somewhat more soluble in cold  $H_2O$ , may be obtained. The total yield is 90-95%.

Other preparative methods:

II. OXIDATION OF TeO2 WITH KMnO4 IN NITRIC ACID SOLUTION:

$$\begin{split} 5 \, \mathrm{TeO}_2 \, + \, 2 \, \mathrm{KMnO}_4 \, + \, 6 \, \mathrm{HNO}_3 \, + \, 12 \, \mathrm{H_2O} \\ &= \, 5 \, \mathrm{H_6TeO}_6 \, + \, 2 \, \mathrm{KNO}_3 \, + \, 2 \, \mathrm{Mn(NO_3)_2}. \end{split}$$

The process is somewhat more cumbersome than method I since it requires, among other things, separate preparation of  $\text{TeO}_2$  as well as repeated recrystallization of the acid to quantitatively remove the simultaneously formed salts. The yield is 75-85%. For detailed description of the procedure, see the literature below.

III. Reaction of Te or TeO<sub>2</sub> with 30% H<sub>2</sub>O<sub>2</sub> in sulfuric acid or alkaline solution and subsequent precipitation of H<sub>6</sub>TeO<sub>6</sub> with concentrated nitric acid.

IV. Oxidation of Te with  $HNO_3$  and  $CrO_3$ . The product must be recrystallized several times in order to remove the  $Cr(NO_3)_3$  byproduct.

PROPERTIES:

Formula weight 229.66. Colorless crystals, stable in air. May occur in a monoclinic modification (space group  $C_{2h}^{5}$ ) and a cubic modification (space group  $O_{h}^{5}$ ); the large crystals are generally monoclinic while the microcrystalline powder frequently consists of a mixture of both forms.

Heating  $H_6 TeO_6$  between 100 and 220°C converts it to solid, water-insoluble polymetatelluric acid; the latter decomposes above 220°C into TeO<sub>3</sub>, and at 400°C or higher into TeO<sub>2</sub> and O<sub>2</sub>. Melts in a sealed tube at about 136°C, forming a concentrated aqueous solution of polymetatelluric acid and some polyorthotelluric acid. d (monoclinic) 3.071; d (cubic) 3.17. Readily soluble in water; very slightly soluble in concentrated nitric acid.

**REFERENCES:** 

- I. J. Meyer and M. Holowatyj, Ber. dtsch. chem. Ges. <u>81</u>, 119 (1948); J. Meyer and W. Franke, Z. anorg. allg. Chem. <u>193</u>, 191 (1930).
- II. F. C. Mathers, C. M. Rice, H. Broderick and R. Forney in: L. F. Audrieth, Inorg. Syntheses, Vol. III, p. 145; New York-London, 1950.
- III. L. I. Gilbertson, J. Amer. Chem. Soc. <u>55</u>, 1460 (1933); A. Gutbier and W. Wagenknecht, Z. anorg. allg. Chem. <u>40</u>, 260 (1904).
- IV. L. Staudenmaier, Z. anorg. allg. Chem. <u>10</u>, 189 (1895).

## Sodium Tetrahydrogentellurate (VI)

#### Na<sub>2</sub>H<sub>4</sub>TeO<sub>6</sub>

 $\begin{array}{rrr} H_{\pmb{6}} TeO_{\pmb{6}} \ + \ 2 \ NaOH \ = \ Na_2 H_4 TeO_{\pmb{6}} \ + \ 2 \ H_2 O \\ 229.66 \ & 80.01 \ & 273.64 \end{array}$ 

A moderately concentrated  $H_8 TeO_8$  solution (p. 451) is mixed at the boiling point with the stoichiometric quantity of concentrated NaOH solution and is kept boiling for a short time. A microcrystalline, granular precipitate of  $Na_2H_4TeO_8$  separates. When the solution has cooled, it is filtered by suction, washed with water at 0°C, and dried in a desiccator over CaCl<sub>2</sub>. The salt thus obtained is analytically pure.

PROPERTIES:

White crystals, stable in air. Decomposes into  $Na_2TeO_3$  when vigorously heated. Very slightly soluble in water.

REFERENCES:

J. Meyer and M. Holowatyj, Ber. dtsch. chem. Ges. 81, 119 (1948).

A. Gutbier, Z. anorg. allg. Chem. 31, 340 (1902).

A. Rosenheim and G. Jander, Kolloid-Z. 22, 23 (1918).

## Sodium Orthotellurate

## Na<sub>6</sub>TeO<sub>6</sub>

Ι.

 $\begin{array}{rcl} H_{6}TeO_{6} &+ & 6 \text{ NaOH} &= & \text{Na}_{6}TeO_{6} &+ & 6 \text{ H}_{2}O \\ 229.66 & & 240.03 & & 361.59 \end{array}$ 

Solid NaOH is melted in a porcelain crucible and the stoichiometric quantity of  $H_{g}TeO_{g}$  is added to the melt at 290-300°C. When the reaction is complete, the crucible is slowly cooled, then broken; alcohol is poured over it and allowed to stand for 4-5 hours. The solution thus formed is filtered hot and slowly concentrated at room temperature until crystallization occurs. After 2-3 days, crystals (1-1.5 cm. long) of  $Na_gTeO_g \cdot 2H_2O$  separate. The product is filtered off, washed with alcohol, and dried in a vacuum desiccator over  $P_2O_5$ . The water of crystallization is given off quantitatively, and the product converts to white, powdery  $Na_gTeO_6$ . The salt thus obtained is analytically pure.

II. Other Preparative Methods: Fusing  $Na_2O$  and  $Na_2TeO_4$  at  $700^{\circ}$ The process requires a large amount of equipment. Detailed descriptions of the apparatus and procedure are given in the literature below.

PROPERTIES:

White, microcrystalline powder; gradually forms  $Na_2H_4TeO_{6'}$ ·  $3H_2O$  in air. Readily soluble in water.

**REFERENCES:** 

- I. J. Meyer and M. Holowatyj, Ber. dtsch. chem. Ges. <u>81</u>, 119 (1948).
- II. E. Zintl and W. Morawietz, Z. anorg. allg. Chem. <u>236</u>, 372 (1938).

# Sodium Telluropentathionate

Na2TeS4O6 · 2 H2O

 $\begin{array}{rrr} {\rm TeO_2} \ + \ 4 \ Na_2S_2O_3 \ + \ 4 \ CH_3COOH \\ {\rm 159.61} & {\rm 632.50} & {\rm 240.21} \end{array}$ 

 $= \underbrace{Na_2 TeS_4O_6}_{397.87} + \underbrace{Na_2S_4O_6}_{270.26} + 4 CH_3 COONa + 2 H_2O \\ 328.16 328.16 36.03$ 

Under the same conditions as those described for the preparation of  $Na_2SeS_4O_6 \cdot 3H_2O$  (p. 434), 110 g. of  $Na_2S_2O_3 \cdot 5H_2O$  is dissolved in 60 ml. of  $H_2O$  and then added over a period of 15 minutes to a solution of 18.8 g. of TeO<sub>2</sub> in 75 ml. of concentrated HCl and 75 ml. of glacial acetic acid. Then 150 ml. of ethanol is added and the mixture is stirred and cooled for 15 additional minutes. Rubbing the walls with a glass rod accelerates the crystallization.

454

The crude product, which contains approximately 25 g. of  $Na_2TeS_4O_6 \cdot 2H_2O$  and 2-4 mole percent of  $Na_2S_4O_6$ , is filtered, washed with ethanol and ether, and dried in vacuum over  $H_2SO_4$ .

To recrystallize the salt, it is dissolved in 60 ml. of 0.2N HCl held below  $45^{\circ}$ C; further procedure is as in the preparation of Na<sub>2</sub>SeS<sub>4</sub>O<sub>5</sub> · 3H<sub>2</sub>O. The yield is 20 g. (45%).

#### PROPERTIES:

Small flakes or flat needles. In large quantities, the crystals appear yellow with an orange cast; individual crystals seem yellow with a greenish cast. Dilute aqueous solutions seem yellow, concentrated ones orange-red. Completely soluble in water but less soluble than the corresponding selenium salt. Aqueous solutions gradually decompose into Te and tetrathionate but the rate of decomposition is smaller than that of the corresponding selenium salt. The solutions can be stabilized by mineral acids, whereas alkalis accelerate the decomposition. Insoluble in ethanol, very slightly soluble in methanol. Readily gives off its water of crystallization over  $H_2SO_4$  in vacuum. The pure material can be stored over  $H_2SO_4$  for several months without decomposition (Te eventually separates).

**REFERENCE:** 

O. Foss in: J. C. Bailar, Inorg. Syntheses, Vol. IV, p. 88, New York-London-Toronto, 1953.

## **Tellurium Sulfur Trioxide**

TeSO<sub>3</sub>

$$Te + SO_3 = TeSO_3$$
  
127.61 80.07 207.68

An excess of molten  $SO_3$  is allowed to react with pure, finely powdered Te at room temperature. The reaction must be protected from moisture. The apparatus and the procedure are the same as for the analogous synthesis of  $S_2O_3$  (p. 380). The product is darkred  $\alpha$ -TeSO<sub>3</sub>; in order to obtain the light brown  $\beta$ -modification, this product is heated for a short time to about  $80^{\circ}C$ .

Stored in ampoules sealed in an oxygen-free atmosphere at pressures below 1 mm. When so stored, it is stable for some time at room temperature. PROPERTIES:

The  $\alpha$ -modification is dark, and the  $\beta$ -form is light brown; very hygroscopic; decomposes on heating above 90°C into SO<sub>2</sub>, TeO<sub>2</sub> and Te.

Reacts violently with water, resulting in the precipitation of Te and formation of  $H_2SO_4$ ,  $H_2SO_3$  and  $H_2TeO_3$ . Soluble in oleum with a blood-red color; insoluble in pure  $SO_3$ .

**REFERENCES:** 

- E. Divers and M. Shimosé, Ber. dtsch. chem. Ges. 16, 1008 (1883).
- A. Damiens, Compt. Rend. Hebd. Séances Acad. Sci. <u>179</u>, 829 (1924).

# SECTION 8

Nitrogen

P. W. SCHENK

#### Nitrogen

 $N_2$ 

The preparation of elemental nitrogen in the laboratory will only occasionally be necessary, for special purposes. Commercially available  $N_2$  in steel cylinders can normally be used when an inert gas is required. In general, no purification is required if the low-oxygen-low-moisture grade is purchased. If the latter is unavailable the gas can be purified by the method described below. Very pure  $N_2$  can be obtained by thermal decomposition of NH<sub>3</sub> or the decomposition of alkali azides.

NITROGEN FROM AZIDES

 $2 \operatorname{NaN}_{3} = 2 \operatorname{Na} + 3 \operatorname{N}_{2}$ 130.0 46.0 67.2 l

A layer (several millimeters thick) of recrystallized and dried NaN<sub>a</sub> is placed in a thin-wall Pyrex decomposition tube a (Fig. 173). The tube is 40 cm. long and 2 cm. in diameter. The lubricated ground glass joint b is kept cool by wrapping it with a wet The entire apparatus is evacuated with a mercury diffusion rag. pump and dried by heating under vacuum. The tube containing the azides is also heated; however, it is not allowed to reach the decomposition temperature of the azide. The internal pressure and the seals of the apparatus are tested with a high-frequency apparatus (vacuum leak tester). Finally, tube a is uniformly heated with an incandescent flame. Stopcock d is then closed and the azide is heated at one spot until decomposition begins. The progress of gas evolution is checked, after closing stopcocks q and  $\mathcal{J}$  and opening d, by observing the pressure increase registered by manometer h. Heating is stopped until the pressure increase slackens. The heating is then resumed until a pressure increase is again recorded on the manometer. When the fine Na dust has settled in flask c, the gas is transferred into flask e. This procedure

is repeated until a sufficient amount of  $N_2$  has been collected. This procedure avoids the use of the rapidly clogging glass-wool plug normally employed to filter out the Na dust. There is no danger of an explosion since a sudden pressure increase would, at worst, break the seal at b and separate the gas generation tube from the ground glass joint. No further purification of the  $N_2$  obtained by this method is necessary.



Fig. 173. Preparation of nitrogen from azides.

The decomposition temperature of NaN<sub>3</sub> is  $275^{\circ}$ C; KN<sub>3</sub> decomposes at  $355^{\circ}$ C; the alkaline earth azides decompose above  $110^{\circ}$ C.

PROPERTIES:

B.p. -195.8°C, m.p. -210°C; d 1.2505 g./liter.

**REFERENCES:** 

E. Tiede, Ber. dtsch. chem. Ges. 46, 4100 (1913); 49, 1745 (1916).
E. Justi, Ann. Phys. [5] 10, 985 (1931).

PURIFICATION OF COMMERCIAL TANK NITROGEN

Water,  $CO_2$  and other impurities (oil vapor) are removed with the usual absorbents. Difficulties are encountered only in the removal of the last traces of  $O_2$ . The common procedure of passing the gas over heated Cu is suitable only when the nitrogen need not be too pure, since the partial pressure of oxygen over glowing CuO cannot be discounted. Thus, one of the following methods should be used.

Purification with "active" copper. Based on Fricke's work, Meyer and Ronge have developed the following practical method. A tube (Fig. 174), 75 cm. long and 4 cm. diameter, is directly wrapped with 10 m. of heating wire with a total resistance of 64 ohms. The inside temperature of the tube is adjusted to about  $170^{\circ}$ C with a small rheostat, using a thermometer as an indicator. This temperature is maintained during both the oxidation and the reduction because recent studies indicate that the apparatus achieves its greatest efficiency and longest life at this temperature. A protective glass jacket, fastened at the ends with asbestos, prevents heat loss and at the same time permits observation of the inside of the tube. The tube is filled according to one of the following procedures.

Either 250 g. of CuCl<sub>2</sub> · 2H<sub>2</sub>O or 366 g. of CuSO<sub>4</sub> · 5H<sub>2</sub>O is dissolved in two liters of water, and 250 g. of purified kieselguhr (boiled with hydrochloric acid and heated to incandescence) is added. The mixture is then precipitated at 60°C with a solution of 200 g. of NaOH in 500 ml. of H<sub>2</sub>O, while vigorously stirred. After ten minutes the mixture is poured into ten liters of distilled water. By allowing to stand, decanting and again suspending in fresh water, the product is washed as well as possible. It is then filtered by suction. The moist cake is then pressed through a die with a high-power screw press (a meat grinder was ineffective). The thin sausages of 4-5 mm. diameter are collected on a sheet of paper. The sausages are dried in air, where they harden and shrink to some extent, and are then broken into pieces 5-10 mm. long. These pieces are dried at 180°C in a drying oven. The dust is sifted off and the pieces are loosely packed into the inner tube of the apparatus. Hydrogen is introduced from the top and heat-



Fig. 174. Removal of oxygen from impure nitrogen ("copper tower"). By using a ground glass joint at s, the tube and the water separator can be made of different glasses.

ing is begun when all the air has been displaced. The water formed during the reduction of the CuO collects in the lower part w of the tube and is removed via the stopcock. The stopcock may be replaced by a ground glass joint, the male part of which may be removed during the reduction. The apparatus is ready for use when the contents have turned deep violet.

Nitrogen is introduced into the apparatus from the top. The gas issuing at the bottom is redried by one of the usual methods.

A second method devised by Meyer and Ronge involves dissolving 120 g. of basic copper carbonate in two liters of concentrated ammonia water. After addition of 420 g. of purified kieselguhr, the solution is evaporated almost to dryness on a water bath. The somewhat moist cake is further treated as indicated above.

Precipitation with NaOH is nevertheless preferable because the CuO is precipitated more evenly and adheres more firmly to the kieselguhr.

The initial reduction of the tube packing takes considerably longer than subsequent regenerations with  $H_2$ .

The particular advantage of the above apparatus is that the activity of the packing can be estimated by visual inspection. As the apparatus is used, one can distinctly see zones of violet Cu, brown CuO and yellow Cu<sub>2</sub>O. This packing can absorb about four liters of O<sub>2</sub>, and thus can be used to purify about 400 liters of N<sub>2</sub> containing 1% O<sub>2</sub>. For continuous operation, two such reactors are prepared, so that one may be regenerated while the other is in use. According to Meyer and Ronge (who also describe the analytical method), nitrogen purified in this manner contains less than  $4 \cdot 10^{-5\%}$  O<sub>2</sub> (see also p. 336).

Purification with ammoniacal copper salt solution. According to H. von Wartenberg, 300 g. of copper shavings is packed into a twoliter steel cylinder and 300 ml. of a mixture prepared from 250 ml. of saturated  $NH_4HCO_3$  solution, 250 ml. of concentrated ammonia, 500 ml. of  $H_2O$  and 100 g. of  $NH_4Cl$  is added. After addition of the solution, the cylinder valve is screwed on and  $N_2$  is introduced to a pressure of 100 atm. gage. Then the steel cylinder is rotated for eight hours on a suitable device (a lathe may be used) or shaken in a shaking apparatus. The  $N_2$  thus obtained need only be purified with  $H_2SO_4$  and dilute KOH, followed by drying with concentrated  $H_2SO_4$ . It is designated as "completely free of oxygen." For determination of oxygen see p. 336.

**REFERENCES:** 

F. R. Meyer and G. Ronge, Z. angew. Chem. 52, 637 (1939).

- H. von Wartenberg, Z. Elektrochem. 36, 295 (1930).
- E. C. Kendall, Science [2] 73, 395 (1931).
- R. Fricke and J. Kubach, Z. Elektrochem. 53, 76 (1949).

# Ammonia

# $\mathbf{NH}_3$

The laboratory preparation of ammonia from ammonium salts should seldom be required since pure synthetic  $NH_3$  is commercially available in steel cylinders. In order to remove minor impurities such as oil vapor, traces of  $CO_2$ , etc., it is sufficient to pass the gas over fresh or well-regenerated activated charcoal.

From time to time, the charcoal is either heated under vacuum or treated with water vapor.

The gas is dried by passing through a series of 0.5-m.-long tubes. The first is filled with soda lime, the second with solid KOH or BaO, and the third with sodium wire. The gas is finally passed over  $P_2O_5$ , since  $NH_3$  predried by the above procedure does not react with  $P_2O_5$ . The absence of this reaction can be used as a criterion for successful predrying. One can also achieve further purification by condensing the gas and dissolving some Na in the condensate. The  $NH_3$  boiling off from the blue solution is completely dry. For suitable apparatus see Part I, p. 86 ff.

For drying, Fehér recommends a small, 1.5-liter steel cylinder with a screw-on lid. Some metallic sodium is added to the cylinder; the cylinder is cooled and filled with the  $NH_3$  (Fig. 175). Since the sodium reacts not only with the water but also gradually with the ammonia,  $H_2$  pressure builds up; this can be observed with a manometer. The  $H_2$  should be vented from time to time.

To obtain a steady supply of purified  $NH_3$ , one can liquefy the gas over very dry  $NH_4NO_3$ . The vapor pressure of the resulting solution is considerably decreased so that it can be stored in a glass ampoule equipped with a stopcock from which the  $NH_3$  can be drawn off as needed.

PROPERTIES:

Gas, very soluble in water. Liquid  $NH_3$  is a solvent resembling water; that is, phenomena observed upon solution of many substances in water (dissociation) are also observed upon their solution in liquid  $NH_3$ . Ammonia itself dissociates to a very small extent into  $NH_4^+$  and  $NH_2^-$  ions. M.p. -77.8°C, b.p. -33.5°C.

REFERENCE:

- F. Fehér, J. Cremer and W. Tromm, Z. anorg. allg. Chem. 287, 175 (1956).
- N<sup>15</sup>H<sub>3</sub> FROM LABELED NH<sub>4</sub>Cl

Ammonia labeled with N<sup>15</sup> is prepared from labeled NH<sub>4</sub>Cl and KOH in an apparatus especially suitable for smaller quantities (Fig. 176). N<sup>15</sup>H<sub>4</sub>Cl (1.2 g.) is introduced via inlet  $\alpha$  into small flask k and mixed with 5 ml. of water. Approximately 4 g. of KOH pellets is introduced via inlet b and about 8 g. via d; the glass lugs prevent their descent. The condensation trap  $f_1$  is cooled in liquid nitrogen after tubes  $\alpha$ , b and d are sealed by fusion, and the pressure is decreased to 450 mm. Stopcock h can remain closed during the gas evolution in spite of the pressure increase, provided the gas



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evolution proceeds in the correct manner; in that case loss of N<sup>15</sup> is impossible. Flask k is carefully heated with a small flame; the salt dissolves and the rising water vapor condenses on the KOH in the neck of the flask. The formation of N<sup>15</sup>H<sub>3</sub> starts as soon as the concentrated alkali begins to flow. The evolving gas carries over enough water vapor to dissolve all the KOH in b, and the reaction is sustained even when the flame is withdrawn. The intermediate condenser c is necessary because otherwise the gas is too moist when it reaches the upper drying tube; even with the condenser, the bottom part of the drying tube (where the gas enters) heats up Some self-regulation is inherent in this simple considerably. arrangement. Too high a rate of gas evolution results in a pressure increase, which in turn increases the solubility of NH<sub>3</sub> in the liquid phase (especially in the water condensed by c) so that the pressure decreases, etc.

The N<sup>15</sup>H<sub>3</sub> evolved in the process may be transferred almost completely to  $f_2$  by boiling for half an hour, shutting of the heat at the end of that period and evacuating the system to 30 mm. The receiver system is sealed off at  $s_2$ , the air is removed with a pump, and  $f_2$  is immersed in liquid nitrogen and  $f_1$  in a Dry Ice-methanol bath. The N<sup>15</sup>H<sub>3</sub> is then distilled from  $f_1$  into  $f_2$ . At the end of the distillation,  $f_2$  is sealed off at constriction  $s_1$ . The dry ammonia may be removed by distillation. The yield is almost 100%.

REFERENCE:

K. Clusius and E. Effenberger, Helv. Chim. Acta 38, 1836 (1955).

## AMMONIA SOLUTION

To prepare analytically pure, carbonate-free ammonia solution,  $NH_3$  from a steel cylinder is passed through a tube packed with activated charcoal and a tube packed with soda lime, and finally into well-boiled water (soda lime can be dispensed with in some instances). The water should not occupy more than 2/3 of the bottle because the dissolution of the  $NH_3$  results in a considerable increase of volume. The outlet tube is closed off by a small soda lime tube. The inlet tube reaches to the bottom of the vessel. Backing up is unlikely because of the high flow rate of the incoming gas. It is advisable to cool the vessel with cold water.

Small quantities of pure, carbonate-free ammonia can easily be prepared by placing a dish of boiled distilled water above a dish of pure concentrated ammonia in a desiccator and allowing to stand overnight. This method of "isothermal distillation" can also be used to prepare other very pure reagents.

PROPERTIES:

Yields  $NH_3$  when boiled; d of a saturated solution at  $15^{\circ}C$ , 0.882.

**REFERENCE:** 

E. Abrahamczik, Chemie 55, 233 (1942).

## Lithium Amide

## LiNH<sub>2</sub>

Metallic lithium is brought into contact with anhydrous liquid  $NH_3$  in a heavy-wall tube. A blue solution of the metal results. When the tube is sealed off and allowed to stand for 2-3 weeks, the  $Li(NH_3)_X$  is converted to  $LiNH_2$  according to the above equation. Because of the pressure increase, the tube must be protected with an iron jacket and the pressure released by opening and closing

the cylinder once or twice. The reaction may also be completed in several hours by heating to  $60^{\circ}$ C.

To prepare larger quantities, Li metal is heated in a trough made of nickel sheeting. This is placed in a glass tube and the assembly is placed in an electric furnace. The tube is inclined and dry  $NH_3$ is passed through from one end while the furnace is heated to 380-400°C. The molten LiNH<sub>2</sub> runs off and solidifies in the cooler portions of the reaction tube. In this manner, fresh surface of the Li metal is continuously exposed.

In another preparatory method, two nickel crucibles may be arranged one above the other in a vertical glass tube placed in an electrical furnace. The bottom of the upper crucible has three 1.5-mm. openings and contains a piece of lithium metal. The furnace is heated to  $400^{\circ}$ C while NH<sub>3</sub> is passed through. The LiNH<sub>2</sub> formed drips into the lower crucible through the holes in the upper. The product is cooled in a stream of NH<sub>3</sub>.

#### PROPERTIES:

Translucent, lustrous, colorless crystal mass. M.p. 380-400°C. Gives off NH<sub>3</sub> on heating under vacuum above  $450^{\circ}$ C, with formation of Li<sub>2</sub>NH. The latter is stable up to 750-800°C, where it decomposes to NH<sub>3</sub> and N<sub>2</sub>. d (17.5°C) 1.178.

**REFERENCES:** 

- A. W. Titherley, J. Chem. Soc. (London) 65, 517 (1894).
- O. Ruff and E. Geisel, Ber. dtsch. chem. Ges. <u>39</u>, 840 (1906); <u>44</u>, 505 (1911).
- R. Juza and K. Opp, Z. anorg. allg. Chem. 266, 313, 325 (1951).

## Lithium Imide

## Li<sub>2</sub>NH

 $\begin{array}{c} 2 \operatorname{LiNH}_2 = \operatorname{Li}_2 \operatorname{NH} + \operatorname{NH}_3 \\ 45.9 \qquad 28.9 \qquad 17.0 \end{array}$ 

Lithium amide, divided into rice-sized grains, is heated slowly to  $360^{\circ}$ C in a glass tube evacuated with a mercury diffusion pump. After 3-4 hours, evolution of NH<sub>3</sub> practically ceases. During the next two hours the temperature is increased to  $450^{\circ}$ C (with continuous pumping). Evacuation of the vessel is continued for an additional two hours at this temperature. The amide should not be allowed to melt during this procedure. PROPERTIES:

d 1.48. Crystallizes in an antifluorite structure. Insoluble in benzene, toluene and ether.

**REFERENCES:** 

O. Ruff and H. Goerges, Ber. dtsch. chem. Ges. 44, 502 (1911). R. Juza and K. Opp, Z. anorg. allg. Chem. 266, 325 (1951).

## Sodium Amide

NaNH<sub>2</sub>

 $Na + NH_3 = NaNH_2 + \frac{1}{2}H_2$ 23.0 17.0 39.0 11.2 l

I. Smaller quantities of NaNH<sub>2</sub> are prepared by moderate heating of Na metal, freed of its crust and oil, in a boat placed in a porcelain or glass tube, through which a stream of NH<sub>3</sub> is passed. The temperature should be kept at about 300°C and the NH<sub>3</sub> must be dried especially carefully. All the air must be removed from the apparatus before the start of the reaction. The end of the reaction is determined by collecting the evolving gases in a test tube over Hg and then immersing the tube in water. The reaction is terminated when no H<sub>2</sub> remains in the test tube after absorption of the NH<sub>3</sub>. The amide is slowly cooled in a stream of NH<sub>3</sub>.

II. Larger amounts of NaNH<sub>2</sub> are prepared by the method of Dennis and Brown, as follows.

A nickel dish containing 100 g. of pure Na metal freed of crust and oil is placed in an iron pot equipped with a lid (Fig. 177). Pure, dry NH<sub>3</sub> gas is introduced as described above. The entire system is heated. When the sodium melts, the end of the inlet tube is immersed in it. The NH<sub>3</sub> stream (from a steel cylinder) should be constant. The apparatus must be provided with a pressure release valve so that sudden plugging will not cause disturbances. The temperature should be kept at approximately  $350^{\circ}$ C and should never be allowed to drop below  $250^{\circ}$ C. After about 5-7 hours the inlet tube is lifted out of the melt, which is tested for completeness of reaction in the manner described above. The product is cooled in a stream of NH<sub>2</sub>.

Still larger quantities of  $NaNH_2$  may be prepared in a carefully cleaned iron retort with polished inside surfaces. The retort should hold about 500 g. of Na. The  $NH_3$  inlet tube should end just short of the surface of the molten metal. Since the reaction proceeds rapidly, especially at the beginning, a sufficiently large and

efficient drying apparatus must be provided for the ammonia. (See p. 461 f., steel cylinder with Na, Fehér's method.) The retort is heated to about  $300^{\circ}$ C. Good seals must be provided to exclude air from the apparatus. During the vigorous uptake of ammonia, especially at the start of the reaction, there is a danger of sucking in air, which may lead to an explosion.



Fig. 177. Preparation of sodium amide. a) Iron pot; b) lid; d) inlet tube; e) outlet tube; f) thermometer; g) nickel dish; h) asbestos stopper; k) asbestos seal; l) tripod.

III. Clusius and Effenberger have described the following procedure for the preparation of NaNH<sub>2</sub> (as well as NaN<sub>3</sub>) with N<sup>15</sup>, using the apparatus shown in Fig. 178. An iron boats is welded to an iron wire, at the end of which there is a slotted sleeve, by means of which the boat is suspended from a protrusion on the ground glass stopper. This prevents creeping of the reaction material. About 10 cm, of clean sodium wire (about 10 mmoles) is charged into the boat. The little vial b contains two drops of concentrated  $H_2SO_4$ ; the storage flask holds the NH<sub>3</sub>. The entire apparatus is evacuated via stopcock  $h_2$ , which is then closed. The NH<sub>3</sub> is admitted to the Na vessel, and the reaction is carried out by heating furnace o to 250-300°C. The course of the reaction is followed through the pressure changes indicated by manometer m. The H<sub>2</sub> formed is from time to time removed by suction via  $h_2$ , while the NH<sub>a</sub> is retained in tube c, which is cooled to  $-180^{\circ}$ C. During the evacuation, the furnace temperature must be temporarily decreased below 210°C (the melting point of NaNH<sub>2</sub>) because the liquid amide dissolves considerable gas and may bump. The reaction is completed in about 24 hours, during which the apparatus must be refilled three times with NH<sub>2</sub>. The remaining gases are removed by suction.

If the final product is to be  $NaN_3$ , dry  $N_2O$  is taken from a previously filled ampoule and reacted with the amide at a furnace temperature of 170-190°C. The reaction requires 36 to 48 hours

for completion. Stopcock  $h_1$  is kept open so that the NH<sub>3</sub> formed via the reaction  $2 \text{ NaNH}_2 + \text{N}_2\text{O} = \text{NaN}_3 + \text{NH}_3 + \text{NaOH}$  may be absorbed by the H<sub>2</sub>SO<sub>4</sub>.

When the reaction is complete, a crust of NaN<sub>3</sub> covers the entire boat. It is dissolved in water, some  $Fe_2O_3$  present is centrifuged off, and the solution is concentrated on the water bath.

IV. Another procedure utilizing liquid  $NH_3$  and Na metal and carried out in the presence of a catalyst yields lower purity  $NaNH_2$  since the catalyst remains in the product; however, the amide is finely divided and free of NaH and unreacted Na.

The catalyst is powdered iron (III) nitrate,  $Fe(NO_3)_3 \cdot 9H_2O$ (1 g. per 100 g. of Na). The reaction is carried out in a threeneck, round-bottom flask equipped with a rugged, tightly sealed stirrer. A spiral condenser which can be cooled with Dry Ice is set in one of the side necks;  $NH_3$  is introduced through the other neck. The flask is half filled with liquid  $NH_3$ , and then the ferric nitrate and approximately 50 g. of Na metal per liter of liquid ammonia are added piece by piece. The sodium pieces are manipulated by spearing them with an iron wire. Whenever the blue solution turns gray, a fresh piece of Na is added. Finally the excess  $NH_3$  is evaporated and the remainder is driven off on a water bath.

PROPERTIES:

White, fibrous crystalline mass. Reacts vigorously with water. M.p.  $210^{\circ}$ C; subl. t. above  $400^{\circ}$ C; dec.  $500-600^{\circ}$ C. When partially oxidized or hydrolyzed by contact with air, often detonates violently upon heating.



Fig. 178. Preparation of sodium amide with  $N^{15}$ . a) ground glass stopper; b) ampoule with  $H_2SO_4$ ; c) condensation trap;  $h_1$ ,  $h_2$ ) stopcocks; m) manometer; o) electric furnace; s) iron boat; t) thermocouple.

**REFERENCES:** 

- A. W. Titherley, J. Chem. Soc. (London) 65, 504 (1884).
- J. Wislicenus, Ber. dtsch. chem. Ges. 25, 2084 (1892).
- L. M. Dennis and A. W. Browne, Z. anorg. allg. Chem. 40, 95 (1904).
- W. C. Johnson and W. C. Fernelius, J. Chem. Ed. <u>6</u>, 443 (1929).
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- K. N. Campbell and B. K. Campbell, Org. Syntheses 30, 72 (1950).

# Hydrazinium Sulfate N<sub>2</sub>H<sub>8</sub>SO<sub>4</sub>

$2 \text{ NH}_{3} +$	NaOCl +	$H_2SO_4$	$= N_2 H_6 SO_4$	+ NaCl	$+ H_2O$
34.1	74.4	98.1	130.1	58.5	18.0

A 1N sodium hypochlorite solution (100 ml.) is added to 200 ml. of 20% ammonia water and 5 ml. of 1% limewater in a one-liter Erlenmeyer flask. The mixture is heated rapidly to boiling and maintained at that temperature for 1/2 hour so that all excess NH<sub>3</sub> is removed and the volume of the solution is reduced to about one half. The solution is then rapidly cooled; sulfuric acid is added to the lukewarm solution until a pH of 7-8 is reached (check with pH paper) and the mixture is left to stand for some time until the gray gelatinous substance settles out. The solution is filtered and the filtrate strongly acidified with sulfuric acid. After standing overnight the hydrazine sulfate is filtered. It is already quite pure but can be further purified by recrystallization from boiling water.

P. Pfeiffer and H. Simons recommend adding Trilon B (sodium ethylenediaminetetraacetate) instead of the limewater. This additive gives a true solution with water and a stable complex with the heavy metals which catalyze the decomposition of monochloramine formed as an intermediate. However, the yield is only 20% and therefore the procedure is of no advantage as of now.

## PROPERTIES:

Shiny, glasslike tablets or prisms. Very slightly soluble in cold water (at  $22^{\circ}$ C, 3.05 g./100 ml. H<sub>2</sub>O); readily soluble in hot water. Insoluble in alcohol. M.p.  $254^{\circ}$ C (dec.).

## **REFERENCES:**

F. Raschig, Ber. dtsch. chem. Ges. <u>40</u>, 4588 (1907).

P. Pfeiffer and H. Simons, Ber. dtsch. chem. Ges. 80, 127 (1947).

#### Hydrazine Hydrate

#### $N_2H_4 \cdot H_2O$

 $\begin{array}{rrr} N_2H_6SO_4 + 2 \text{ KOH} = N_2H_4 \cdot H_2O + K_2SO_4 \\ 130.1 & 112.2 & 50.0 & 174.3 \end{array}$ 

Dry hydrazine sulfate (100 g.) is mixed with an equal amount of powdered KOH in a Cu or Ag retort, 15 ml. of  $H_2O$  is added and the hydrazine hydrate formed is distilled through a downward inclined glass condenser. Heating is almost unnecessary at the beginning, but considerable heat eventually must be supplied in order to bring the reaction to completion. The hydrazine hydrate, which still contains water at this point, is purified by fractional distillation. Pure hydrazine hydrate distills between 117 and 119°C. The first cut is converted to hydrazine sulfate. The yield is 10 g. of hydrazine hydrate from 100 g. of hydrazine sulfate (about 25% of theoretical).

#### PROPERTIES:

Strongly refracting liquid; fumes in air; not very mobile. Peculiar odor which is, however, unlike ammonia. Miscible with water and alcohol, but not with ether, chloroform and benzene. B.p. (739.5 mm.)  $118.7^{\circ}$ C.

## Hydrazine

#### $N_2H_4$

Several procedures are available for the preparation of anhydrous hydrazine. Of these, the dehydration of  $N_2H_4 \cdot H_2O$  and the cleavage of hydrazine salts with  $NH_3$  are of special interest. The first method yields anhydrous  $N_2H_4$ , if an efficient column with about 15 theoretical plates is used. The distillation is carried out in a stream of  $N_2$ , using a mixture of 100 g. of 78.5% hydrazine hydrate and 140 g. of NaOH. The  $N_2H_4$  obtained is at least 99.5% pure.

According to Fehér, the cleavage of hydrazine salts with NH<sub>3</sub> proceeds as follows:

I.

$$N_{2}H_{4} \cdot HCl + NH_{3} = N_{2}H_{4} + NH_{4}Cl$$
  
68.5 17.0 32.0 53.5

The reaction tube r (Fig. 179) made of Pyrex (outside diameter 45 mm., length 1200 mm.) is heated in an aluminum jacket equipped with a heating coil. The temperature is measured by thermocouples inserted between the glass tube and the furnace wall.

If the starting material is laboratory-made hydrazine dihydrochloride, about 500 g. of the moist compound may be charged in such a way that the upper third of the tube remains free. The charge is heated to about 120 to 150°C, and a fast stream of dry air is drawn through the tube by means of the pump. When the salt is dry and no moisture is seen in the connecting tubes,  $h_2$  is closed and the temperature increased to 190°C. In the vacuum created by an aspirator, the hydrazine dihydrochloride loses about half its HCl over a period of 2-3 hours and is converted to the monochloride. (If commercial monochloride is used as the starting material, this part of the operation may be omitted.) The temperature of the reaction tube is now decreased to 160°C and dry ammonia (see p. 461) is bled in through the three-way stopcock  $h_{1}$ . As soon as atmospheric pressure is attained, the apparatus is connected to the flowmeter m via  $h_2$ , and the ammonia stream regulated so that a pressure somewhat in excess of atmospheric exists inside the apparatus ( $h_2$  may have to be closed a little). Traps  $f_1$  and  $f_2$  are cooled to  $-30^{\circ}$ C. The course of the reaction can be followed by changes in the pressure, the NH<sub>3</sub> flow rate, the temperature of the



Fig. 179. Preparation of large quantities of anhydrous hydrazine. a) Tap;  $\mathcal{J}_1$  high-efficiency condensation trap;  $\mathcal{J}_2$  condensation trap; m ) flow meter; r ) reaction tube;  $s_1$  ) air cooling coil;  $s_2$  ) spiral condenser; v ) manometer for pressure control.

furnace, and the rate of condensation of the  $N_2H_4$ . The reaction is terminated after 20-30 hours. About 125 ml. of  $N_2H_4$  is collected in the first trap, from which it can be removed via *a* even during the run, using a two-neck, round-bottom flask. The dissolved  $NH_3$  is separated by refluxing at 100°C in a  $H_2$  stream. The yield is 90%. After vacuum distillation to separate any hydrochloride carried over, the  $\rm N_2H_4$  is 100% pure.

A special apparatus for the one-step preparation of larger quantities of  $N_2H_4$  has been described by Fehér, Cremer and Tromm.

A simplified, smaller apparatus, capable of only modest yields (about 30%) is shown in Fig. 180. Hydrazine monochloride (two moles) is placed in a 500-ml., two-neck, round-bottom flask. Tubes  $r_2$  and  $r_3$  are closed off with rubber tubing and pinch clamps. The apparatus is evacuated via  $r_1$  by means of an aspirator. A clear melt free of bubbles is formed on heating to 190°C on an oil bath. The temperature is then decreased to about 150°C and  $h_1$  is closed. Dry NH<sub>3</sub> gas is introduced through  $r_2$  and trap f is cooled to -10 to -30°C. A drying tube filled with KOH pellets is connected at  $r_3$  and the NH<sub>3</sub> stream is adjusted to 50-100 bubbles per minute. As the melt becomes viscous, the temperature is increased gradually to 190°C. The reaction is complete when further flow of



Fig. 180. Simplified apparatus for the preparation of small amounts of anhydrous hydrazine:  $r_1$  is connected to an aspirator;  $r_2$  is connected to an ammonia cylinder; f is a condensation trap.

 $NH_3$  becomes impossible because of solidification of the melt. The  $N_2H_4$  is freed of  $NH_3$  as described above (refluxing in a stream of  $H_2$  or  $N_2$ ).

Π.

$$\begin{array}{r} N_2H_4 \cdot H_2SO_4 + 2 NH_3 = N_2H_4 + (NH_4)_2SO_4 \\ 130.1 & 34.0 & 32.0 & 132.1 \end{array}$$

This procedure, based on the insolubility of  $(NH_4)_2SO_4$  in liquid NH<sub>3</sub>, has been worked out by Fehér et al., as well as Glemser, and yields N<sub>2</sub>H<sub>4</sub> of about 99.5% purity from commonly available hydrazine sulfate.

ANALYSIS:

By the method of Penneman and Audrieth; titration with 0.1 M potassium iodate solution in 3-5N HCl.

PROPERTIES:

Oily, strongly fuming liquid. Vigorously attacks cork, rubber and other organic substances. M.p. 1.8°C, b.p. 113.5°C. Explosive if ignited or overheated, especially in the presence of air. Miscible with water and alcohols. Only slightly soluble in other solvents.

**REFERENCES:** 

- F. Raschig, Ber. dtsch. chem. Ges. 43, 1427 (1910).
- W. Schlenk and T. Weichselfelder, Ber. dtsch. chem. Ges. <u>48</u>, 669 (1915).
- F. Fehér, J. Cremer and W. Tromm, Z. anorg. allg. Chem. <u>287</u>, 175 (1956).
- R. A. Pennemann and L. F. Audrieth, Anal. Chem. 20, 1058 (1948).
- L. F. Audrieth and B. Ackerson, in Ogg: The Chemistry of Hydrazine, New York, 1951.
- G. Pannetier and R. de Hartoulary, Bull. Soc. Chim. France. Mem.
  (5) <u>21</u>, 941 (1954), Battelle Development Corp., Swiss Pat. 291184 (1950-1953); Mathieson Chem. Corp., Brit. Pat. 703 150 (1950-1954).
- O. Glemser, H. Weber and H. Duyster, Z. anorg. allg. Chem. 286, 205 (1956).
- F. Fehér, Private communication.
- H. Bock, Z. anorg. allg. Chem. 293, 264 (1958).

# Hydrazoic Acid

HN<sub>3</sub>

$$\begin{array}{l} N_{a}N_{3} + H_{2}SO_{4} = HN_{3} + N_{a}HSO_{4} \\ 65.0 \quad 98.1 \quad 43.0 \quad 120.1 \end{array}$$

A solution of NaN<sub>3</sub> is mixed with a small amount of litmus and sulfuric acid (2:1 diluted with  $H_2O$ ) is slowly added. When an excess of acid is present, the mixture is slowly distilled. By repeated fractionation one obtains 91% acid, which can be made anhydrous by distillation over CaCl<sub>2</sub>. However, extraordinarily violent explosions sometimes occur with this procedure. According to Günther and Meyer, HN<sub>3</sub> can be prepared in a relatively safe fashion by replacing the sulfuric acid with stearic acid. Pure NaN<sub>3</sub> is mixed in a round-bottom flask with stearic acid; a trap cooled to  $-40^{\circ}C$  is fused directly to this flask. The reaction flask is evacuated and heated. The HN<sub>3</sub> is then purified by distillation at -50 to  $80^{\circ}C$ .

## PROPERTIES:

Water-clear liquid. B.p.  $37^{\circ}$ C, m.p.  $-80^{\circ}$ C. Very mobile and extremely explosive. However, even concentrated solutions can be handled without too much danger. Unbearably pungent odor. Inhalation of the vapor causes dizziness, headache and strong irritation of the mucous membranes.

#### REFERENCES:

L. M. Dennis and H. Isham, J. Amer. Chem. Soc. 29, 27 (1907).
P. Günther and R. Meyer, Z. Elektrochem. <u>41</u>, 541 (1935).
J. Einig, German Patent 435 654, from Chem. Zentr. <u>1926</u> II, 3072.
W. Hoth and G. Pyl, Angew. Chem. <u>42</u>, 888 (1929).
A. W. Browne, J. Amer. Chem. Soc. <u>27</u>, 551 (1905).
J. Martin, J. Amer. Chem. Soc. <u>49</u>, 2133 (1927).
L. F. Audrieth, Chem. Rev. <u>15</u>, 169 (1934).

## SOLUTIONS OF HYDRAZOIC ACID

Because of the very high danger of explosion with pure  $HN_3$ , it is expedient to work only with its relatively harmless solutions.

I. Aqueous solution: A solution of NaN<sub>3</sub> (15 g.) and NaOH (5 g.) is prepared in 150 ml. of water in a 250-ml. distillation flask equipped with an addition funnel and a high-efficiency condenser. The end of the condenser is connected by means of an adapter to a 500-ml. suction flask, which contains 100 ml. of water. The suction outlet of the flask is connected to a tube leading directly to the hood. The contents of the flask are heated to boiling (very important!), and 90 ml. of 40%  $H_2SO_4$  is added dropwise. The distillation is continued until about 50 ml. of the solution remains in the flask. In this manner one obtains a solution containing about 3%  $HN_3$  (0.6-0.7N). The initial addition of NaOH is a precautionary measure which definitely precludes too high a concentration of  $HN_3$  in the cold solution.

II. Anhydrous Ether Solution: Since the distribution of  $HN_3$  between water and ether is approximately 1:7, one can extract an aqueous solution of  $HN_3$  with ether. However, even in this case it is preferable to use a distillation method:  $NaN_3$  (30 g.) is dissolved in 100 ml. of water, 150 ml. of ether is added, and the mixture is placed in a 500-ml., round-bottom flask. The latter is equipped with an adapter fitted to a condenser, followed by a suitable icecooled receiver flask containing 100 ml. of ether. The roundbottom flask is also equipped with an addition funnel, the tip of which is immersed in the liquid and through which 30 ml. of concentrated  $H_2SO_4$  is slowly added. The bulk of the ether and  $HN_3$  distill off during the addition of the  $H_2SO_4$ . The remainder is driven off by heating on a steam bath. The ether distillate is dried over  $CaCl_a$  and then distilled from this desiccant.

**REFERENCES:** 

- W. S. Frost, J. C. Cothran and A. W. Browne, J. Amer. Chem. Soc. 55, 3516 (1933).
- L. F. Audrieth and C. F. Gibbs in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London 1939, p. 77.

## Azides

SODIUM AZIDE, NaN3

 $2 \text{ NaNH}_{2} + \text{ N}_{2}\text{O} = \text{ NaN}_{3} + \text{ NaOH} + \text{ NH}_{3}$ 78.0 44.0 65.0 40.0 22.4 l

Sodium azide is prepared in the same apparatus used for the preparation of sodium amide (see Fig. 177, p. 459). Following the preparation of sodium amide, an N<sub>2</sub>O generator (from ammonium nitrate) is attached instead of the ammonia generator (see Fig. 179, p. 470). The N<sub>2</sub>O outlet tube is equipped with a glass tee immersed in mercury. It acts as a pressure release valve should the inlet tube plug. The water, formed together with the N<sub>2</sub>O, is collected in the receiver (Fig. 184). Finally, the gas is well dried over soda lime and sodium hydroxide before reaching the reactor (Fig. 177). The N<sub>2</sub>O inlet tube may not dip into the sodium azide melt since NaN<sub>3</sub> is solid at the reaction temperature and thus would plug the tube. About five hours are required for the conversion of 25 g. of NaNH<sub>2</sub> to NaN<sub>3</sub>. The crude product obtained must be recrystallized from water unless it is used for the preparation of HN<sub>3</sub> or other azides.

Other methods of preparation:  $N_2H_4 + HNO_2 = HN_3 + 2H_2O$ . Hydrazine hydrate (5 g.) is dissolved in 50 ml. of absolute ether, the solution cooled with ice, and 37.5 ml. of 4N sodium methoxide solution and 12.6 ml. of ethyl nitrite are added. The solution is allowed to stand for a while in the ice and is then slowly warmed to room temperature. The NaN<sub>3</sub> precipitates and is washed, after suction filtration, with a methanol-ether mixture. If hydrazine hydrate is unavailable, the corresponding quantity of hydrazine sulfate can be used. It is ground with the methoxide solution, and the sodium sulfate precipitated is filtered off. After the addition of ether, the solution so obtained is reacted with ethyl nitrite. One then proceeds as described above.

Another procedure is to dissolve 26 g. of hydrazine sulfate in 140 ml. of sodium hydroxide solution (containing 28 g. of NaOH),

add 22 ml. of ethyl nitrite, and shake the mixture for six hours in a pressure bottle. The unreacted ethyl nitrite is purged with air, and the alcohol is similarly removed on a water bath. The solution is used directly for the preparation of  $HN_{3}$ .

For the procedure for the preparation of N<sup>15</sup>-labeled NaN<sub>3</sub>, see p. 466.

**REFERENCES:** 

L. M. Dennis and A. W. Browne, Z. anorg. allg. Chem. <u>40</u>, 95 (1904). W. Wislicenus, Ber. dtsch. chem. Ges. <u>25</u>, 2084 (1892).

K. Clusius and E. Effenberger, Helv. Chim. Acta 38, 1834 (1955).

LITHIUM AZIDE, LiN3

With gentle warming, NaN<sub>3</sub> (91.30 g.) and  $LiSO_4 \cdot H_2O$  (1.41 g.) are dissolved in 7 ml. of  $H_2O$ . Then 35 ml. of 96% alcohol is added with shaking. The solution is filtered after ten minutes and the residue of Na<sub>2</sub>SO<sub>4</sub> and LiSO<sub>4</sub> washed with alcohol. The filtrate and wash solutions are evaporated on a water bath and dried in a drying oven at 80°C. This crude product is digested at 35°C for two minutes with 10 ml. of 96% alcohol and filtered, and the solution is dried as described above. The yield is 0.6 g of 99.5% LiN<sub>3</sub>.

**REFERENCE:** 

N. Hofmann-Bang, Act. Chem. Scand. 11, 581 (1957).

ALKALI AZIDES FROM CARBONATES

 $\begin{array}{ll} Na_2CO_3 \; (K_2CO_3; Rb_2CO_3; Cs_2CO_3) + \; 2\; HN_8 \; = \\ 106.0 & (138.2; \;\; 231.0; \;\; 325.8) & 86.1 \end{array}$ 

 $= 2 \operatorname{NaN}_{3} (\operatorname{KN}_{3}; \operatorname{RbN}_{3}; \operatorname{CsN}_{3}) + H_{2}O + CO_{2}$ 130.0 (162.2; 255.0; 349.9) 18.0 44.0

According to Suhrmann and Clusius, the required quantity of  $HN_3$  is prepared in the following manner. The NaN<sub>3</sub> and the calculated amount of 6%  $H_2SO_4$  are placed in a 300-ml., round-bottom Pyrex flask equipped with a ground glass stopper. A distillation tube, sufficiently long to prevent spraying over, is fused laterally to the neck of the flask. The tube end is immersed in a Pt dish filled with an alkali carbonate solution (for the preparation of pure alkali carbonates see p. 987) so that the HN<sub>3</sub> which

comes over is completely absorbed. In order to ensure complete conversion of the carbonate to the azide, an excess of  $HN_3$  is used. After the reaction is complete, the alkali azide solution is evaporated on a water bath until the onset of crystallization. Crystallization on cooling is carried out without disturbing the solution to prevent the inclusion of mother liquor. After standing for several hours, the crystalline paste is separated from the mother liquor by suction filtration in a Pt Gooch crucible and washed with small quantities of distilled water. The crystals are dried in a drying oven at about  $80^{\circ}$ C and stored in a desiccator over  $P_2O_5$ . The stoppers of the flask and the lid of the desiccator are not greased in order to avoid contamination of the preparation.

PROPERTIES:

NaN<sub>3</sub>: Formula weight 65.02. Decomposes at 275°C without melting. Solubility (17°C) 41.7 g./100 g. H<sub>2</sub>O; (16°C) 0.315 g./100 g. absolute alcohol; insoluble in ether. d (x ray) 1.838. F5<sub>1</sub> structure type.

KN<sub>3</sub>: Formula weight 81.12. M.p.  $343^{\circ}$ C, decomp. t.  $355^{\circ}$ C. Solubility (water) (0°C) 41.1 g., (17°C) 49.6 g., (100°C) 105.7 g./100 g. H<sub>2</sub>O; (alcohol, 16°C) 0.137 g./100 g. absolute alcohol; insoluble in ether. d (x ray) 2.045. F5<sub>2</sub> structure type.

 $RbN_3$ : Formula weight 127.50. M.p. 321.°C, decomp. t. 395°C (in a quartz tube). Solubility (16°C) 107.1 g./100 g. H<sub>2</sub>O; 0.182 g./ 100 g. absolute alcohol; insoluble in ether. d 2.788. Probably F5<sub>2</sub> structure type.

CsN<sub>3</sub>: Formula weight 174.93. M.p.  $326^{\circ}$ C, decomp. t.  $390^{\circ}$ C (in a quartz tube). Solubility ( $16^{\circ}$ C)  $307.4 \text{ g}./100 \text{ g}. H_2O$ ; 1.037 g./ 100 g. absolute alcohol; insoluble in ether.

**REFERENCE:** 

R. Suhrmann and K. Clusius, Z. anorg. allg. Chem. 152, 52 (1926).

Azides of Be, Mg, B, Al, Ga, Si: Their preparation is described by E. Wiberg and H. Michaud, Z. Naturforschg. <u>9 b</u>, 495 (1954); see also section on Alkaline Earth Metals.

# Chlorine Azide

 $Cl N_3$ 

$$\begin{array}{r} NaN_3 + Cl_2 = N_3Cl + NaCl \\ 65.0 \quad 70.9 \quad 77.5 \quad 58.4 \end{array}$$

I. A solution of chlorine azide in  $CCl_4$  is prepared by mixing a solution of  $NaN_3$  in water with a solution of NaOCl. The reaction follows the above equation. A layer of  $CCl_4$  is introduced underneath the NaOCl and the mixture is acidified with boric or acetic

acid while stirring vigorously. The two layers are then separated in a separatory funnel.

II. A gaseous mixture of  $ClN_3$  and  $N_2$  is obtained by mixing solutions containing one mole each of  $NaN_3$  and NaOCl and dropping the mixture slowly from an addition funnel into a boric acid or dilute acetic acid solution. Simultaneously, a stream of  $N_2$  is passed into the flask through a tube, the end of which is immersed in the solution. This stream immediately removes the  $ClN_3$  formed. This procedure has proven safer than that in which the gas is withdrawn using a vacuum, in which explosions may occur. It is advisable to feed the  $ClN_3$  diluted with  $N_2$  directly into nonaqueous solvents such as alignatic hydrocarbons, chloroform or methanol.

Other procedures: A solution of  $ClN_3$  in ether may be prepared by introducing  $Cl_2$  into a dispersion of  $AgN_3$  in ether and filtering off the AgCl. Bromine azide can be similarly prepared.

PROPERTIES:

Condenses at  $-15^{\circ}$ C to an orange-colored liquid. Solidifies at about  $-100^{\circ}$ C to a yellow, very explosive mass. The melting and boiling points have not been accurately determined because of its explosiveness.

**REFERENCES:** 

F. Raschig, Ber. dtsch. chem. Ges. 41, 4149 (1908).

W. J. Frierson, J. Kronrad and A. W. Browne, J. Amer. Chem. Soc. 65, 1696, 1698 (1943).

## Monochloramine

Cl NH<sub>2</sub>

$$\begin{array}{rrr} 2\,NH_{3}\,+\,Cl_{2}\,=\,Cl\,NH_{2}\,+\,NH_{4}Cl\\ 34.l & 70.9 & 51.5 & 53.5 \end{array}$$

I. The calculated amount of  $Cl_2$  (but not an excess!) required for conversion to NaOCl is introduced into 250 ml. of 2N NaOH with efficient cooling. Several pieces of ice are dropped into the solution (precooled to 0°C), and 250 ml. of 1N NH<sub>4</sub>OH (also precooled) is then added at once. The mixture is then placed in a distillation flask equipped with an ice-cooled spiral condenser. Using an aspirator vacuum, 50 to 60 ml. is distilled off at 30 to 40°C. The chloramine solution thus obtained is stable for some time at 0°C. However, the HCl formed during its decomposition further accelerates that process. Thus, a small excess of ammonia should be maintained in the solution. In order to obtain the monochloramine from its solution, the latter is placed in a distillation flask to which a drying tube filled with ignited  $K_2CO_3$  is attached. A trap cooled with liquid nitrogen is attached to this drying tube. After evacuation, the flask with the solution is slightly heated. The potassium carbonate removes the water vapor carried over and the monochloramine collects in the trap. The accompanying ammonia is removed by suction at  $-60^{\circ}C$ .

It should be noted that  $NCl_3$ , which causes violent explosions, is easily formed by decomposition during this operation.

II. According to Sisler, Neth, Drago and Yaney, gaseous ClNH, is prepared as follows. As shown in Fig. 181, the reaction vessel is a glass tube a (65 cm. long and 50 mm. in diameter) which is closed at both ends with rubber stoppers. The right end of the tube is half filled with glass wool, packed a little more tightly near the stopper. The left stopper is perforated and has five 8-mm.-diameter tubes. Chlorine gas, diluted with  $N_2$ , is introduced through the middle tube. A glass rod d, sealed in with rubber tubing c, is used to remove any plugs of NH<sub>4</sub>Cl. The NH<sub>3</sub> gas is introduced via the four tubes arranged symmetrically around the central tube. The four tubes are slightly bent toward the center. Gaseous Cl., N. and NH, are withdrawn from steel cylinders, dried and metered into the tube in a ratio of 1:3:30. The Cl<sub>2</sub> flow rate should be about 0.01-0.05 mole/hour. A large excess of NH<sub>3</sub> and a reasonably high rate of flow are essential. The exit gases can be condensed in liquid nitrogen.



# Fig. 184. Preparation of monochloramine. a) Reaction tube, half filled with glass wool; d) glass rod for removal of NH<sub>4</sub>Cl plugs.

The preparation of  $NH_{g}Cl$  solution in liquid  $NH_{3}$  has been described by J. Jander, Z. anorg. allg. Chem <u>280</u>, 264 (1955).

PROPERTIES:

M.p.  $-66^{\circ}$ C. Colorless crystals which decompose very readily.

**REFERENCES:** 

- H. Markwald and W. Wille, Ber. dtsch. chem. Ges. 56, 1319 (1923).
- G. H. Coleman and H. L. Johnson, in H. S. Booth, Inorganic Syntheses, Vol. I, New York-London 1939, p. 59.
- R. Mattair and H. H. Sisler, J. Amer. Chem. Soc. 73, 1619 (1951).
- H. H. Sisler, F. T. Neth, R. S. Drago and D. Yaney, J. Amer. Chem. Soc. 76, 3906 (1954).

## Nitrogen Trichloride

NCl<sub>3</sub>

 $\frac{NH_3 + 3 Cl_2}{17.0 \ 212.7 \ 120.4 \ 109.4} + 3 HCl_{109.4}$ 

Preparation in solution: A 10 to 20% solution of CCl<sub>4</sub> in CHCl<sub>3</sub> (225 ml.) is placed in a one-liter round-bottom flask and 600 ml. of 10%  $(NH_4)_2SO_4$  solution is added. The flask is closed with a cork stopper equipped with gas inlet and outlet tubes. The outlet tube is connected to a flask containing alkali to absorb excess Cl<sub>2</sub>. The inlet tube is then connected to a Cl<sub>2</sub> cylinder or a Cl<sub>2</sub> generator. The quantity of chlorine generated from 60 g. of KMnO<sub>4</sub> and 300 ml. of concentrated hydrochloric acid (about 20 liters = 70 g.) is introduced while vigorously shaking the flask. During this operation the solution becomes warm, but if it was well prechilled, further cooling is usually unnecessary. The two layers are now separated; the CHCl<sub>3</sub> layer is washed by repeated and vigorous shaking for five minutes with 5%  $(NH_4)_2SO_4$  solution. Finally the layers are separated as well as possible and the CHCl<sub>3</sub> solution is dried with CaCl<sub>2</sub>. The solution contains about 12% NCl<sub>3</sub>.

PROPERTIES:

The solution may be stored for several days in the dark under  $(NH_4)_2SO_4$  solution. Before use it is separated (after shaking the two solutions vigorously) and dried as described. Handling of such solutions is not dangerous up to a concentration of 18% NCl<sub>3</sub>.

**REFERENCES:** 

W. A. Noyes, J. Amer. Chem. Soc. 50, 2902 (1928).

- W. C. Bray and C. T. Dawell, J. Amer. Chem. Soc. <u>39</u>, 896, 905 (1917).
- W. A. Noyes and A. B. Haw, J. Amer. Chem. Soc. <u>42</u>, 2167 (1920).
# Nitrogen Triiodide

 $3 I_2 + 5 NH_3 = NI_3 \cdot NH_3 + 3 NH_4 I$ 761.5 85.2 411.8 434.9

A suspension of very finely divided iodine is prepared by pouring an alcoholic iodine solution into water. The solution is allowed to settle and is filtered. The precipitate is washed with water, mixed with 0°C concentrated aqueous ammonia, and allowed to stand. The ammonia is decanted and renewed several times. The product is then sucked dry on filter paper and washed several times with alcohol and finally with ether. The filter is removed from the funnel while still very wet with ether; it may not be touched after it dries in air since even the slightest contact causes the nitrogen trilodide to detonate immediately. Preparations made in this manner often contain less  $NH_3$  than indicated by the above formula, but it is impossible to obtain pure  $NI_3$ .

For lecture demonstrations the explosive nitrogen triiodide is prepared more simply by precipitating a  $KI_3$  solution with concentrated ammonia and washing the precipitate with alcohol and ether on a suction filter.

PROPERTIES:

Brown-black powder. Explodes upon the slightest touch. Detonation occurs occasionally even under water. Storage is thus impossible.

**REFERENCES**:

Mallet, Chem. News 39, 257 (1879).

J. Szuhay, Ber. dtsch. chem. Ges. 26, 1933 (1893).

O. Ruff, Ber. dtsch. chem. Ges. 33, 3025 (1900).

J. Jander and E. Schmid, Z. anorg. allg. Chem. 292, 178 (1957).

### Nitrogen Tribromide and Monobromamine

To prepare bromine-nitrogen compounds such as  $NBr_3 \cdot 6NH_3$ and  $BrNH_2$ , see the original papers by M. Schmeisser, Z. anorg. allg. Chem. <u>246</u>, 284 (1941) and J. Jander, Z. anorg. allg. Chem. 296, 117 (1958).

# Thionyl Imide

SONH

 $\begin{array}{c} \text{SOCl}_2 \,+\, 3\,\text{NH}_3 \,=\, \text{SONH} \,+\, 2\,\text{NH}_4\text{Cl} \\ 119 \quad 51 \quad 63 \quad 107 \end{array}$ 

The reaction shown above takes place very readily in the gaseous phase if an excess of  $NH_3$  is carefully avoided. The slightest excess

immediately causes polymerization to red-colored products. The apparatus shown in Fig. 182 is used.



Fig. 182. Preparation of thionyl imide. b) Bodenstein valve; c) needle valve; f condensation trap; m ) Hg manometer; r) reaction vessel;  $s_1, s_2$ ) flowmeters;  $v_1, v_2$ ) surge vessels.

Ammonia dried over Na is stored in ampoule a. It can be transferred from a into reaction vessel r via precision needle value c. The flow meter  $s_2$ , the Hg manometer m, and the surge flask  $v_2$  (volume about one liter) are used to adjust and maintain the rate at which the gas enters the reaction vessel. A similar apparatus is used to maintain a stoichiometric ratio of SOC1, to NH, and a constant flow of thionyl chloride vapor from d. Because  $SOCl_2$ attacks stopcock lubricants, Hg and metals, it is advisable to use a glass control and shut off valve b of the type developed by Bodenstein (see Part I, p. 62). The two flowmeters are filled with bromonaphthalene. They are calibrated with SOCl<sub>2</sub> and NH<sub>3</sub> prior to the experiment. The gas, at a given pressure and flow rate, is condensed in f. The quantity accumulated over a measured time is titrated to yield the calibration. The preparation of SONH proceeds as follows: Initially, one obtains the desired SOC1, flow rate by adjusting b. The SOCl<sub>2</sub> reservoir is kept at a constant temperature by an ice bath around it. The constant temperature in the NH<sub>3</sub> reservoir is similarly maintained and the corresponding flow rate is established. The flow rates must be such that only a very small excess of SOCl<sub>2</sub> is maintained. The trap f is then cooled with liquid nitrogen. (Naturally, it is essential to place at least one, and preferably two traps cooled with liquid nitrogen in front of the mercury pump used to evacuate the entire apparatus.) A colorless condensate of SONH collects rapidly in the trap. If the tube connecting the trap and the reaction vessel is about 10 mm. in diameter, about 1 g. of SONH collects within a few minutes.

PROPERTIES:

Colorless liquid. M.p.  $-85^{\circ}$ C. Polymerizes rapidly at the melting point to a clear glasslike mass which soon discolors to a yellow-red, brown and then cloudy substance.

**REFERENCE:** 

P. W. Schenk, Ber. dtsch. chem. Ges. 75, 94 (1942).

# Sulfamide

### $SO_2(NH_2)_2$

$$\begin{array}{rl} \mathrm{SO_2Cl_2} + 4 \mathrm{NH_3} = 2 \mathrm{NH_4Cl} + \mathrm{SO_2(NH_2)_2} \\ \mathrm{135.0} & 68.1 & \mathrm{107.0} & 96.1 \end{array}$$

Well-dried NH<sub>3</sub> (300-400 ml.) (see p. 461) is liquefied in the reactor shown in Fig. 183. A large diameter KOH drying tube is attached to the side of the reaction vessel; the latter is closed off with a two-hole stopper. A stirring rod is inserted through one of the holes and a cylindrical addition funnel is inserted through the other. A solution of 100 ml. of SO<sub>2</sub>Cl<sub>2</sub> in 100 ml. of petroleum ether (b.p. 40°C) is added over a period of 3-4 hours from the funnel. The reaction vessel is kept at  $-80^{\circ}$ C. The solid crust of NH<sub>4</sub>Cl and imidosulfuryl compounds forming on the surface of the liquid NH. must be broken up with the stirring rod. After all the SO<sub>2</sub>Cl<sub>2</sub> has reacted, the product is poured into a large porcelain dish. The excess NH<sub>3</sub> evaporates overnight and the remaining material is placed in a round-bottom flask. The rest of the NH<sub>3</sub> and the petroleum ether are evacuated over a period of two hours by means of an aspirator. The white, solid crude product is dissolved in about 400-500 ml. of water to which 5 ml. of concentrated hydrochloric acid per 100 ml. of solution has been added. To hydrolyze the major part of the chain-type imido compounds formed, the solution is rapidly heated to 70-80°C and kept at this temperature for ten minutes. The solution is allowed to cool to 30°C and then evaporated under vacuum at this temperature until dry. If the product is still somewhat moist, it must be dried in a vacuum desiccator over  $H_2SO_4$ . To separate the sulfamide, the dry crude product is extracted at room temperature two or three times with 400 ml. of acetone. Evaporation of the acetone solution yields 40-60 g. of almost pure sulfamide, which can be further purified by recrystallization from ethanol or ethyl acetate.

PROPERTIES:

Colorless rhombic plates. M.p. 93°C. Readily soluble in water and hot ethanol, very slightly soluble in cold ethanol. Tasteless. Upon heating decomposes according to:  $3 \text{ SO}_2(\text{NH}_2)_2 =$  $(\text{SO}_2\text{NH})_3 + 3 \text{ NH}_3$ . Forms a silver salt with AgNO<sub>3</sub> only upon addition of NH<sub>3</sub>. The silver salt, SO<sub>2</sub>(NHAg)<sub>2</sub>, is very slightly soluble in water.



Fig. 183. Reactor for the preparation of sulfamide.

**REFERENCES:** 

M. Traube, Ber. dtsch. chem. Ges. <u>25</u>, 2427 (1892); <u>26</u>, 610 (1893). A. Hantzsch and A. Holl, Ber. dtsch. chem. Ges. <u>34</u>, 3430 (1901). W. Appel, Private communication.

# Trisulfimide and Its Silver Salt

 $(SO_2NH)_3$   $(SO_2N)_3Ag_3 \cdot 3H_2O$ 

 $3 \text{ SO}_{2}(\text{NH}_{2})_{2} = (\text{SO}_{2}\text{NH})_{3} + 3 \text{ NH}_{3}$   $288.3 \qquad 237.2 \qquad 51.1$ 

 $(SO_2NH)_3 + 3 AgNO_3 + 3 H_2O = (SO_2N)_3Ag_3 \cdot 3 H_2O + 3 HNO_3$ 237.2 509.7 54.0 611.9 189.0

Pure sulfamide (m.p.  $93^{\circ}C$ ; 2 g.) is placed in a heavy-wall test tube. A thermometer is introduced so that the mercury bulb is completely covered with sulfamide. A side arm of the reaction tube, lightly closed with cotton, is used as a pressure equalizer. The tube is slowly heated to  $92^{\circ}C$  by insertion to a depth of 5 cm. in an oil bath. When the sulfamide has melted, the temperature is slowly raised to  $180^{\circ}C$  over a period of one hour. During the second hour, the temperature is increased to  $200^{\circ}C$  and maintained there for four hours. After a total of six hours of heating, the reaction vessel is removed from the oil bath and allowed to cool. The cold, fused cake of trisulfimide is dissolved in about 20 ml. of water; the solution is then diluted with 500 ml. of boiling water. A solution of 4 g. of AgNO<sub>3</sub> in 50 ml. of water is added. After brief boiling, the small amount of coagulated brownish precipitate is filtered off in a heated funnel and the clear filtrate is left to cool overnight. The fine, long needles of the silver sulfimide which separate are washed with cold water and dried in air. The yield is 2.8-3.0 g. of (SO<sub>2</sub>N)<sub>3</sub>Ag<sub>3</sub>·3 H<sub>2</sub>O, corresponding to 66-71% of theoretical. Further purification (usually superfluous) is accomplished by recrystallization from boiling distilled water.

PROPERTIES:

(SO<sub>2</sub>NH)<sub>3</sub>: Formula weight 237.25. M.p. 165°C. (SO<sub>2</sub>N)<sub>3</sub>Ag<sub>3</sub>. 3 H<sub>2</sub>O: Formula weight 611.91. Needle-shaped crystals. Loses its water of crystallization at 110°C.

REFERENCE:

G. Heinze and A. Meuwsen, Z. anorg. allg. Chem. 275, 49 (1954).

# Nitrous Oxide

 $NH_4NO_3 = N_2O + 2H_2O$ 22.4l

80

36

### N<sub>9</sub>O

Ι.

II.

Very pure NH<sub>4</sub>NO<sub>3</sub> is completely dehydrated in adrying oven at 160 to 170°C and the melt is allowed to solidify in a desiccator. After pulverization it is placed in a flask provided with a sealedon trap. A heating coil is wrapped around the neck of the flask in order to prevent the water formed during the reaction from condensing and flowing back into the hot melt (Fig. 184). The condensation trap b is ice-cooled and is used to retain the major portion of the water formed. The flask is very carefully heated over a wire mesh. The reaction starts at 170°C and is exothermic. The temperature may not exceed 250°C, lest N<sub>2</sub> and NO evolve. Sudden heating and too large charges should be avoided, since the reaction can become explosive due to its exothermicity. The gas is scrubbed by passage through a 50% potassium hydroxide solution; the traces of  $O_2$  may be removed by washing with an alkaline dithionite solution. According to Manchot, the often recommended washing with concentrated FeSO<sub>4</sub> solution is without merit.

$$\begin{array}{c} {\rm K_2SO_3(NO)_2} = {\rm K_2SO_4} + {\rm N_2O} \\ {\rm 218.3} & {\rm 174.3} & {\rm 22.4} \ l \end{array}$$

A suspension of 15 g. of K<sub>2</sub>SO<sub>3</sub>(NO)<sub>2</sub> in 150 ml. of H<sub>2</sub>O is prepared. A trace of alkali is added to the water to retard premature

484

decomposition of the salt. Complete dissolution of the salt is not necessary. Dilute sulfuric acid is added dropwise through an addition funnel. The reaction vessel should be cooled when large charges are used. The product gas is pure after washing with 4N KOH.

III. Pure  $N_2O$  in steel cylinders is commercially available for anesthetic purposes. Further purification of this gas (or of the gas prepared by the above two methods) can be effected by liquefaction and fractional distillation at low temperatures.



Fig. 184. Preparation of nitrous oxide.

PROPERTIES:

Formula weight 44.02. Colorless gas. M.p.  $-90.6^{\circ}$ C, b.p.  $-88.5^{\circ}$ C. At room temperature, the solubility in water corresponds approximately to that of CO<sub>2</sub>.

**REFERENCES:** 

- H. Gehlen, Ber. dtsch. chem. Ges. 65, 1130 (1932).
- H. L. Johnston and H. R. Weiner, J. Amer. Chem. Soc. <u>56</u>, 625 (1934).
- H. L. Johnston and W. F. Giauque, J. Amer. Chem. Soc. <u>51</u>, 3194 (1929).
- V. Meyer, Lieb. Ann. Chem. 175, 141 (1875).
- P. Baumgarten, Ber. dtsch. chem. Ges. 71, 80 (1938).
- W. Manchot, M. Jahrstorfer and H. Zepter, Z. anorg. allg. Chem. <u>141</u>, 48 (1924).

### Nitric Oxide

NO

I.

Sulfuric acid (50%) is added dropwise to a solution of 30 g. of KNO<sub>2</sub> and 15 g. of KI in 100 ml. of  $H_2O$ . The gas is washed with

90% sulfuric acid and 50% potassium hydroxide solutions, dried by passing through a U tube cooled with Dry Ice-ether mixture, and condensed over P<sub>2</sub>O<sub>5</sub> by means of liquid nitrogen. The most volatile fraction is allowed to escape and the middle fraction is redistilled. Instead of the expensive KI,  $K_4 Fe(CN)_6$  may be used.

$$K_{4}Fe(CN)_{6} + KNO_{2} + 2 CH_{3}COOH =$$

$$(3 H_{2}O) \qquad 85.1 \qquad 120.1$$

$$422.4$$

$$= K_{3}Fe(CN)_{6} + 2 CH_{3}COOK + NO + 4 H_{2}O$$

$$329.3 \qquad 198.1 \qquad 22.4 l \qquad 72.1$$

A mixture of 9g. of KNO<sub>2</sub> and 100 ml. of a solution of  $K_4$  Fe(CN)<sub>6</sub> in H<sub>2</sub>O (saturated at room temperature) is prepared; dilute acetic acid is added dropwise. The liberated gas is purified with 50% KOH solution, solid KOH and P2O5. Additional purification is accomplished by distillation, as above.

II. FROM NITROSYL HYDROGEN SULFATE AND MERCURY

Mercury is added dropwise from an addition funnel to (NO)HSO4 or a mixture of NaNO2 (2-4 g.) and concentrated H2SO4 (100 ml.). The Bodenstein drip funnel described in Part I, p. 78 and Fig. 57, is preferred. This apparatus permits the NO to evolve easily under vacuum. The gas is purified by methods indicated above.

III. FROM NITRITE AND DILUTE SULFURIC ACID

 $6 \operatorname{NaNO}_2 + 3 \operatorname{H}_2 \operatorname{SO}_4 = 4 \operatorname{NO} + 2 \operatorname{H}_2 \operatorname{O} + 3 \operatorname{Na}_2 \operatorname{SO}_4 + 2 \operatorname{HNO}_3$ 294.2 89.6 l 36.0 426.1126.0 414.0

A layer of NaNO, is covered with two to three times its weight of water, and dilute sulfuric acid is added dropwise. Rod-shaped solid nitrite can also be decomposed with dilute sulfuric acid in a Kipp generator. Caution is recommended because the riser of the generator is easily plugged by the precipitated Na<sub>2</sub>SO<sub>4</sub>. The liberated gas is washed with 4N KOH; it is quite pure. It may be purified further as above.

IV. Another simple and successful procedure is as follows. Alternate layers of 8.5 g. of finely ground  $FeSO_4 \cdot 7H_2O$  and a finely pulverized mixture of 8.5 g. of NaBr and 4 g. of fused NaNO2 are placed in a dry, 100-ml. round-bottom flask. Mixing of the

layers by shaking the flask initiates the reaction, which is accompanied by foaming. The yield is about 1 liter of gas from the above quantity of starting materials. The gas contains 98.8% NO and 1.2% N<sub>2</sub>.

Other preparative possibilities: From copper and nitric acid. Gas prepared in this manner is rather impure.

The often recommended purification of NO by absorption in  $FeSO_4$  solution, followed by boiling, does not appear to be effective. Purification with strong alkali is likewise of doubtful efficacy. In both cases, contamination with  $N_2O$  may occur.

PROPERTIES:

Formula weight 30.01. Colorless gas, slate-blue liquid; the solid condensed directly from the gaseous phase is a white, fluffy snow. M.p.  $-163.7^{\circ}$ C, b.p.  $-151.8^{\circ}$ C. The only substance which may be used as a sealing liquid for storage of the gas is mercury. Oxidizes very readily in air.

**REFERENCES:** 

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# Nitrogen Trioxide

### $N_2O_3$

Since  $N_2O_3$  is in equilibrium with NO and  $NO_2$  according to the equation  $N_2O_3 \neq NO + NO_2$ , it can readily be prepared from these compounds. The following method is also very convenient:

I.  $2 \text{ HNO}_3 + A_{S_2O_3} + 2 \text{ H}_2\text{O} = 2 \text{ H}_3\text{AsO}_4 + N_2\text{O}_3$ 126.0 197.8 36.0 283.8 76.0

Nitric acid (d 1.30-1.35) is added dropwise to pulverized  $As_2O_3$ . The reaction is initiated by slight heating (70°C).

# II. FROM NITROSYL HYDROGEN SULFATE

 $2(\text{NO})\text{HSO}_4 + \text{H}_2\text{O} = 2\text{H}_2\text{SO}_4 + \text{N}_2\text{O}_3$ 254.2 18.0 196.2 76.0

Since pure (NO)HSO<sub>4</sub> is not required, the stock solution is prepared by the addition of SO<sub>2</sub> to fuming  $HNO_3$  until a thick liquid forms. Water is then added dropwise to this paste in a suitable reactor, such as a flask equipped with a ground glass joint and a dropping funnel.

### PROPERTIES:

Green to blue liquid. Does not have a uniform boiling point but first loses NO, which results in an increased boiling point. M.p.  $-103^{\circ}$ C. Soluble in benzene, CCl<sub>4</sub>, toluene and CHCl<sub>3</sub> with a blue coloration.

**REFERENCES:** 

- G. Lunge, Ber. dtsch. chem. Ges. 11, 1229, 1641 (1878).
- W. Biltz, W. Fischer and E. Wünnenberg, Z, anorg. allg. Chem. <u>193</u>, 355 (1930).
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# Nitrogen Dioxide

### NO<sub>2</sub>, N<sub>2</sub>O<sub>4</sub>

# $\frac{Pb(NO_3)_2}{331.2} = \frac{PbO_2}{239.2} + 2 \frac{NO_2}{92.0}$

Pulverized lead nitrate is dried for several days in a drying oven at 110 to  $120^{\circ}$ C and placed in tuber (Fig. 185) made of high-melting glass. The tube is heated in a slow O<sub>2</sub> stream until gas evolution begins. The major portion of the moisture is frozen out in trap 5 at -15 to  $-20^{\circ}$ C. Additional purification is obtained in horizontal U tubes 6 and 7 over PbO<sub>2</sub> and P<sub>2</sub>O<sub>5</sub>, respectively. The gas is liquefied in vessel 8 at  $-78^{\circ}$ C. Stopcock *h* is then closed and distillation into vessel 9 is carried out using an aspirator vacuum. The first fraction is removed by suction, and after the contents freeze, the tube is fused at *a*. The gas is ultimately distilled into individual ampoules 10, which are then sealed.

For a method of preparing very pure  $NO_2$  from  $N_2O_5$  see L. Hackspill and Besson, Bull. Soc. Chim. France, Mém. (5) <u>16</u>, 479 (1949).



Fig. 185. Preparation of nitrogen dioxide. r) Tube of high melting glass; a and b) seal-off points; 1) mercury valve;
2) concentrated H<sub>2</sub>SO<sub>4</sub>; 3, 4) soda lime; 5) ice-salt-cooled trap; 6) PbO<sub>2</sub>; 7) P<sub>2</sub>O<sub>5</sub>; 8, 9) condensation traps; 10) storage ampoules; 11) drying agent.

PROPERTIES:

Formula weight 46.01. Brown, extremely poisonous gas. Absorbed by alkali, forming nitrite and nitrate. M.p.  $-10.8^{\circ}$ C, b.p. 21.2°C.

**REFERENCES**:

M. Bodenstein, Z. phys. Chem. 100, 68 (1922).

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- F. E. C. Scheffer and C. P. Treub, Z. phys. Chem. 81, 308 (1913).
- A. Klemenc, Die Behandlung and Reindarstellung von Gasen [Treatment and Purification of Gases], Vienna, 1948, p. 207.

### Nitrogen Pentoxide

 $N_2O_5$ 

$$\begin{array}{c} 2 \text{ HNO}_3 + P_2 O_5 = N_2 O_5 + 2 \text{ HPO}_3 \\ 126.0 & 142.0 & 108.0 & 160.0 \end{array}$$

Nitric acid, as concentrated as possible (d1.525), is cooled with an ice-salt freezing mixture and carefully dehydrated with excess  $P_2O_5$ , using adequate cooling and small batches of acid. The mixture is then distilled very slowly from a retort — preferably in a stream of oxygen containing some ozone. The apparatus must not contain any stoppers or connections made of organic material. The yield is about 80 g. of  $N_2O_5$  from 150 g. of HNO<sub>3</sub>.

Cäsar and Goldfrank recommend freezing 70-80 ml. of highly concentrated HNO<sub>3</sub> (d 1.5) contained in a large, three-neck, round-bottom flask cooled with Dry Ice mixture (see Fig. 186). An O<sub>2</sub> stream containing ozone is passed through the flask and 100-125 g. of  $P_2O_5$  is added at once through the central tube. The tube is closed immediately with a glass stopper. The mixture is then allowed to thaw slowly. A room temperature water bath may be used to accomplish this. Large quantities of  $N_2O_5$  mixed with  $N_2O_4$  are rapidly evolved and are condensed in the large trap b, which is cooled to  $-78^{\circ}$ C. When most of the product (recognizable by its reddish vapors) has been transferred, the flask is shaken, and when the renewed evolution of gas again decreases, the flask is heated for several hours in a stream of  $O_2$  containing  $O_3$ . When all the product gas has been transferred and condensed, the trap is removed and the condensate fractionated by vacuum distillation.



Fig. 186. Preparation of nitrogen pentoxide. a) Three-neck round-bottom flask with a ground glass stopper;b) condensation trap with Dry Ice bath.

Other preparative methods: From  $AgNO_3$  and  $Cl_2$  or from  $AgNO_3$  and  $POCl_3$ . It is best to introduce ozone-containing oxygen into the liquid  $N_2O_4$ . For example, 5 g. of liquid  $N_2O_4$  is placed in a U tube cooled in an ice-salt mixture. Oxygen containing 6-7% ozone is passed through the tube for one hour.

### **PROPERTIES:**

Colorless crystals which decompose slowly at room temperature and are moderately stable only below  $10^{\circ}$ C. The melting point has not been determined. Subl. t.  $34^{\circ}$ C. According to R. Schwartz,  $N_2O_5$  reacts with  $H_2O_2$  with the formation of pernitric acid:  $N_2O_5 + H_2O_2 = HNO_4 + HNO_3$ .

**REFERENCES:** 

W. Biltz, W. Fischer and E. Wünnenberg, Z. anorg. allg. Chem. <u>193</u>, 360 (1930).

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### Nitric Acid

### HNO<sub>3</sub>

When necessary, purification of technical-grade nitric acid in the laboratory is accomplished by distillation over a small amount of  $AgNO_3$ . The first and last fractions are discarded; the resulting acid is free of halogens.

Several methods have been suggested for the preparation of anhydrous nitric acid.

I. Acid of the highest possible concentration is distilled; then an inert, dust-free gas stream, preferably preheated, is passed through to remove the nitrogen oxides. It is ultimately distilled over  $P_2O_5$ . Analysis then indicates whether the acid contains free  $N_2O_5$ . If so, some dilute acid, containing enough water to dissolve the  $N_2O_5$  present, is added.

II. By distillation with  $H_2SO_4$ . The starting material is acid of the highest possible concentration; the apparatus shown in Fig. 187 is used. A 600-ml. glass flask is equipped with a ground glass joint with a boiling capillary a and a ground glass thermometer b. An efficient condenser is attached at c and terminates in a receiver. The apparatus is connected to an aspirator via a safety trap. The





three-way stopcock also permits attaching a manometer. The distillation flask is filled with 150 ml. of  $HNO_3$ , 300 ml. of  $H_2SO_4$ ,

both precooled in an ice-salt mixture. Cooling should be continued during the mixing process. The apparatus is then evacuated and carefully heated on a water bath. Colorless  $HNO_3$  passes into the receiving vessel at 22 mm. and  $37-40^{\circ}C$ . The condensate is redistilled in the same manner, using twice the volume of concentrated  $H_2SO_4$ . At 20 mm. the pure acid distills over between 36 and  $38^{\circ}C$ . The ground glass joints should obviously not be lubricated with any organic material. If a sealing agent is necessary, some  $P_2O_5$  or  $H_2SO_4$  may be used.

It is also recommended that a stream of oxygen containing some ozone be passed through the highly concentrated acid and that it be distilled under aspirator vacuum.

PROPERTIES:

Formula weight 63.02. B.p.  $83^{\circ}$ C (increases after some boiling to  $87^{\circ}$ C because of decomposition), m. p.— 41°C. After dilution with twice the amount of water, nitric acid not containing nitrogen oxides does not discolor KMnO<sub>4</sub>. d<sup>15</sup><sub>4</sub> 1.522. Pure, 100% acid cannot be kept without decomposition for a very long time, especially in the light. Aqueous HNO<sub>3</sub> has a boiling point maximum at 121°C (68% HNO<sub>3</sub>).

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# Hyponitrous Acid

# $H_2N_2O_2$

$$\begin{array}{c} \mathrm{Ag_2N_2O_2} + 2\,\mathrm{HCl} = \mathrm{H_2N_2O_2} + 2\,\mathrm{AgCl} \\ \mathrm{275.8} & \mathrm{72.9} & \mathrm{62.0} & \mathrm{286.7} \end{array}$$

Ether, dehydrated over Na wire, is saturated with dry HCl, cooled and treated with  $Ag_2N_2O_2$  until the yellow color of the latter persists. Complete exclusion of atmospheric moisture is required. The solution is rapidly filtered through a dry filter

into a crystallization dish placed in a desiccator. To accomplish this, the addition funnel is connected with a tube in the lid of a desiccator provided with a side vent. A small dish with KOH pellets and another with a small amount of concentrated  $H_2SO_4$  are put in the desiccator. The desiccator is then rapidly evacuated. Hyponitrous acid separates immediately.

PROPERTIES:

Colorless, flakelike crystals. Very explosive, detonates even when rubbed with a glass rod. Ignites with solid KOH. Sometimes decomposes spontaneously, seemingly without cause. Readily soluble in alcohol; less soluble in ether, chloroform and benzene; very slightly soluble in ligroin.

**REFERENCES:** 

A. Hantzsch and L. Kaufmann, Lieb. Ann. Chem. <u>292</u>, 317 (1896); Ber. dtsch. chem. Ges. 29, 1394 (1896).

# Silver Hyponitrite

### $Ag_2N_2O_2$

I.	2 HONHSO <sub>3</sub> Na +	2 KOH	$= K_2 N_2 O_2 +$	$2  NaHSO_3$	$+ 2 H_2 O$
	270.2	112.2	138.2	208.1	36.0

Sodium hydroxylamine monosulfonate is prepared from hydroxvlamine disulfonate. This salt is obtained in solution according to the instructions given on p. 503. However, the salt is not isolated from its solution by addition of KCl. Instead, the solution is evaporated under vacuum until the major portion of the Na<sub>2</sub>SO<sub>4</sub> (formed together with the monosulfonate) precipitates. At this point 1 ml. of the solution should react completely with about 10-12 ml. of 1N iodine solution. When this is achieved, 100 ml. of solution is mixed in a one-liter flask with 100 g. of solid KOH and heated for two hours on a water bath at 60 to 70°C. A thick paste of bisulfite and hyponitrite is formed. A test is performed to determine whether an added suspension of HgO is still reduced to Hg. [Half a gram of the paste is diluted with water and a HgO suspension added. The HgO is obtained by precipitation of a Hg(NO<sub>3</sub>)<sub>2</sub> solution with a small excess of sodium hydroxide solution and is purified by repeated decantation with water.] The paste is then diluted to three liters with water and treated, while shaking vigorously, with HgO until the color of the latter persists. A little steatite powder is added to facilitate the ensuing filtration. The reaction mixture

is filtered, the filtrate is diluted to twice its volume, and 0.25 liter of the diluted filtrate is put aside. A 1% AgNO<sub>3</sub> solution is added slowly and with stirring to the main batch. The precipitate at first redissolves but the yellow Ag<sub>2</sub>N<sub>2</sub>O<sub>2</sub> eventually settles. The addition of AgNO<sub>3</sub> solution is discontinued upon the appearance of a persistent, dark precipitate. This precipitate is then removed by addition of the solution previously put aside. A total of 6-7 liters of AgNO, solution is required. The suspension is left to settle in a glass cylinder, and the mother liquor is then siphoned off. The precipitate is washed several times, with the supernatant liquor removed by decantation and siphoning. (The excess Ag can be recovered from the mother liquor and the first washings with HCl.) The siphoning process is repeated 8 to 10 times. The precipitate is then suction-filtered, washed with water and dried at 100°C. Light should be avoided as much as possible during all the operations with the silver salt. If the silver salt is to be further processed, it is best stored in paste form.

II.

 $2 \operatorname{NaNO_2}_{138.0} + 4 \operatorname{Na}_{2} + 2 \operatorname{H_2O}_{2} + 2 \operatorname{AgNO_3}_{39.8} =$   $4 \operatorname{Rg_2N_2O_2}_{2} + 2 \operatorname{NaNO_3}_{3} + 4 \operatorname{NaOH}_{275.8} = 170.0 = 160.0$ 

Sodium amalgam (2500 g. of Hg + 16 g. of Na) is added from a funnel over a period of forty-five minutes to a vigorously stirred, chilled solution of 20 g. of NaNO<sub>2</sub> and 10 g. of NaOH in 100 ml. of H<sub>2</sub>O. The stirring is discontinued, the supernatant decanted from the Hg, and the solution almost completely neutralized with dilute HNO<sub>3</sub>. (The solution should remain slightly alkaline and the temperature should not exceed 0°C.) Mercury (II) oxide is then added until it begins to settle unchanged on the bottom. The mixture is then filtered and precisely neutralized with dilute HNO<sub>3</sub>. The prepared solution of AgNO<sub>3</sub> is then added immediately. The yellow precipitate of Ag<sub>2</sub>N<sub>2</sub>O<sub>2</sub> is suction-filtered, washed with hot water, dissolved in very dilute, ice-cold HNO<sub>3</sub>, and precipitated with ammonia. The purification by precipitation may be repeated. The yield is about 6-7 g.

E. Abel and J. Proisl describe an electrolytic method of preparation.

PROPERTIES:

Yellow powder, stable in the absence of light. Darkens on exposure to light.

**REFERENCES:** 

- A. Hantzsch and F. Kaufmann, Lieb. Ann. Chem. 292, 320 (1896).
- F. Raschig, Schwefel- und Stickstoffstudien [Studies on Sulfur and Nitrogen], Leipzig-Berlin 1924, p. 94.
- E. Abel and J. Proisl, Mh. Chem. 72, 1 (1939).

# Sodium Hyponitrite

### $Na_2N_2O_2 \cdot 9 H_2O$

I.

 $2 \text{ NaNO}_3 + 8 \text{ Na} + 4 \text{ H}_2\text{O} = \text{Na}_2\text{N}_2\text{O}_2 + 8 \text{ NaOH}$ 170.0 184.0 72 106.0 320.0

Sodium amalgam is added with stirring or shaking to an icecooled solution of 85 g. of NaNO3 in 250 ml. of H2O. The amalgam is prepared by dissolving 58 g. of Na in 4000 g. of Hg. When three quarters of the amalgam has been added, the cooling is discontinued and the remainder of the amalgam is added at once. Shaking is continued for 10-15 minutes, during which the temperature increases to  $40^{\circ}$ C. When the temperature begins to fall the mixture is poured into a closed flask with a narrow neck and the first flask is rinsed with 2-3 ml. of H<sub>2</sub>O. The washings are combined with the main solution and the whole shaken vigorously for about ten minutes. To determine whether all the hydroxylamine formed has disappeared, a drop of the solution is mixed with a little water and AgNO<sub>3</sub>. The solution is decanted from the Hg and placed over  $H_2SO_4$  in a vacuum desiccator at 35 to 40°C in order to remove all the NH<sub>3</sub>. The Na<sub>2</sub>N<sub>2</sub>O<sub>2</sub> • 9H<sub>2</sub>O separates during this operation. It is suction-filtered on a fritted glass filter, washed with alcohol at a temperature above 10°C to remove traces of NaOH, and dried in a desiccator.

II. 
$$2 \operatorname{Na} + 2 \operatorname{NO} = \operatorname{Na_2N_2O_2}_{46.0 \quad 60.0 \quad 106.0}$$

A flask equipped with gas inlet and outlet tubes is connected, via a tee, with an  $H_2$  generator on the one side and an NO generator on the other side (see p. 485). The flask is charged with 0.3 g. of Na metal (in small pieces), 90 g. of pyridine distilled over KOH, and 30 g. of thiophene-free benzene dried over Na wire. The air is displaced with  $H_2$ , and the flask left overnight under the pressure of the  $H_2$  generator. By the next day the Na metal disintegrates to a loose pulp of the green-black pyridine compound. The outlet tube is opened and NO is slowly admitted. By the time the bulk of the  $H_2$  is displaced, the absorption of NO becomes so violent that the apparatus must be cooled. The absorption takes place chiefly during the first ten minutes and is practically terminated after the first 30 minutes, even though less than half of the theoretical amount is absorbed during that time. The NO is then displaced with H<sub>2</sub>, and the product is rapidly suction-filtered on a large filter and thoroughly washed with dry ether. The Na left over is removed mechanically. The light-brown mass is dissolved in a small amount of water. Treatment of the brown solution with several times its volume of alcohol yields a precipitate of Na<sub>2</sub>N<sub>2</sub>O<sub>2</sub>, at first in the form of an oil and finally as shiny, almost white flakes. The flakes are suction-filtered and washed with a small amount of alcohol and ether. Repetition of the procedure yields snow-white plates. The yield is 4.5 g. of Na<sub>2</sub>N<sub>2</sub>O<sub>2</sub> • 9H<sub>2</sub>O.

Other preparative methods: Sufficient NaCl solution is added to a suspension of  $Ag_2N_2O_2$  so that a faint yellow color persists. The mixture is filtered and washed briefly, and the solution is evaporated in vacuum until it crystallizes. It is dried as above.

PROPERTIES:

Formula weight  $(Na_2N_2O_2 \cdot 9H_2O)$  268.14. Small granular crystals or plates. Loses water of crystallization under vacuum. In its anhydrous form, it resembles hydrated magnesium carbonate. Readily soluble in  $H_2O_*$ .

**REFERENCES:** 

- E. Divers, J. Chem. Soc. (London) 75, 96 (1899).
- F. Raschig, Schwefel- und Stickstoffstudien [Studies on Sulfur and Nitrogen], Leipzig-Berlin 1924, p. 95.
- E. Weitz and W. Vollmer, Ber. dtsch. chem. Ges. 57, 1015 (1924).

### Nitramide

### NH<sub>2</sub>NO<sub>2</sub>

$$NO_{2}NHCOOH = NH_{2}NO_{2} + CO_{2}$$
  
106.0 62.0 44.0

a) The potassium nitrocarbamate required for the synthesis is prepared as follows: 100 g. of potassium cyanate is dissolved in the minimal amount of warm, 50% alcohol and the clear solution is added to an excess of alcoholic hydrogen chloride (96% alcohol). After 24 hours, the solution is neutralized with  $BaCO_3$  and suction-filtered, and the bulk of the alcohol is evaporated. The remaining portion is dried in a vacuum desiccator and the residue is extracted

with ether. The ether solution is dried and distilled. The yield is about 66 g. of the ethyl ester of carbamic acid (plates, m.p. 49-50°C, b.p. 180°C). The 66 g. of the ester is added with vigorous stirring to 170 ml. of cold, pure, NO<sub>2</sub>-free H<sub>2</sub>SO<sub>4</sub>. Upon completion of the addition and subsequent temperature drop to below 0°C, 37 g. of ethyl nitrite is added at once. The temperature must drop to below  $-5^{\circ}$ C within ten minutes. The lower the temperature, the better the yield. After standing for 45 minutes at a temperature below -5°C, the mixture is poured over 0.5 to 0.75 kg. of ice and extracted four times with 150 ml, of ether. The solution is dried and diluted to one liter, and dry NH<sub>3</sub> is introduced. The precipitate that separates is suction-filtered and air-dried on clay. The yield is 30-35 g. (The yield can be improved by working up the mother liquor and by further treatment of the aqueous acid solution with ether.) A solution of 25 g, of this salt in 100 ml, of water is prepared and cooled to 0°C. Then 750 ml. of methanolic potassium hydroxide (250 g. of KOH/liter), precooled to  $0^{\circ}$ C, is added with vigorous stirring, which results in a temperature rise. The mixture is ice cooled, but not continuously, so the temperature can be held at  $7^{\circ}$ C. Should local gas evolution occur, the solution is immediately and vigorously stirred. The solution is left to crystallize and is suction-filtered after about one hour. The crystals are thoroughly washed with alcohol and dried in a desiccator. The yield is 20-25 g. Although potassium nitrocarbamate is decomposed by water, it can be stored in a desiccator.

b) A total of 25 g. of potassium nitrocarbamate is introduced in small portions into a mixture of 25 ml. of concentrated  $H_2SO_4$ and 200 g. of ice, and the solution is saturated with  $(NH_4)_2SO_4$ . The mixture is then extracted 30 times with ether until an evaporated sample no longer evolves gas upon addition of a drop of ammonia. The ether solution is put in a flask and evaporated by a dry air stream at room temperature. The residue is dissolved in a small quantity of absolute ether, and the nitramide is precipitated with ligroin. It is washed with petroleum ether and dried on clay. The lengthy ether extraction can be avoided by superimposing a layer of ether on the aqueous solution and solidifying the latter, while stirring, by cooling it with Dry Ice-acetone mixture. The ether is then decanted. Just four repetitions of this process result in transfer of 80% of the nitramide to the ether.

The following procedure is much more rapid, but the yield is also much lower. Potassium nitrocarbamate (25 g.) is added, with shaking, to a mixture of 10 ml. of concentrated  $H_2SO_4$ , 100 g. of ice and 200 ml. of ether. The reaction mixture is cooled in an icesalt bath. Approximately 300 g. of ignited  $Na_2SO_4$  is added portionwise, with shaking and cooling until the mass solidifies. The mixture is left to stand for two hours with occasional kneading, the ether is removed by suction, the mixture is kneaded again using 200 ml. of ether, which is then removed, and the procedure is repeated a third time. The  $NH_2NO_2$  is obtained from the ether solution by the method described above. The yield is about 3-4 g.

PROPERTIES:

Shiny white plates from ligroin. M.p.  $72-75^{\circ}C$  (dec.). Soluble in ether, alcohol, water and acetone; less soluble in benzene; almost insoluble in ligroin. It can be obtained with especial ease by precipitation with chloroform from ether solutions.

**REFERENCES:** 

- J. Thiele and A. Lachmann, Ber. dtsch. chem. Ges. <u>27</u>, 1909 (1894); Lieb. Ann. Chem. <u>288</u>, 273, 297 (1895).
- C. A. Marlies and V. K. LaMer, J. Amer. Chem. Soc. <u>57</u>, 2008 (1935).
- I. N. Brönstedt and K. Pedersen, Z. phys. Chem. 108, 187 (1924).
- O. Folin, J. Amer. Chem. Soc. 19, 341 (1897).
- C. A. Marlies, V. K. LaMer and J. Greenspan, in H. S. Booth, Inorganic Syntheses, Vol. I, New York-London 1939, p. 68.

# Hydroxylammonium Chloride

# (NH<sub>3</sub>OH)Cl

### I. PREPARATION BY ELECTROLYTIC REDUCTION OF HNO3

A lead sheet cylinder is placed inside a cylindrical filtering jar (appr. 10 cm. I. D.) (Fig. 188). The inside cylinder should contact the jar walls as uniformly as possible (a lead sheet vessel of similar size may be used instead of the cylinder). The lead sheet is very carefully cleaned with a soft wire brush and thoroughly amalgalgamated with Hg(NO<sub>3</sub>), solution and Hg. A porous clay cell is inserted; this serves as the anode chamber. The anode placed in this chamber consists either of a water-cooled lead coil or a lead vessel closed on top by a lid equipped with an inlet and an outlet for cooling water. The entire vessel is immersed in ice water, and the anode cooling water is also kept as cold as possible with ice water. A ring-shaped stirrer which can be moved up and down about once a second by a simple mechanical device (a string attached eccentrically to a disk suffices) is placed around the anode. The anode chamber is first filled with 50% sulfuric acid. After the clay cell has been well soaked with acid, the outer annulus is filled with the same acid, the stirrer is put into operation, and the current is adjusted to 24 amp. A solution of 20 g, of nitric acid diluted with 30 ml. of  $H_2O$  is added to the cathode liquid over a period of two hours from a funnel equipped with a capillary tip.

The temperature should not exceed 15°C. The electrolysis is continued for approximately another 45 minutes until the cathode liquid gives only a faint test for  $NO_3$  (FeSO 4 and concentrated  $H_2SO_4$ ). The cathode liquid is then removed from the reaction vessel and diluted with an equal quantity of  $H_2O$ . All  $SO_4^{2-}$ present is precipitated with the calculated stoichiometric quantity of warm BaCl, solution, using intermittent cooling. The precipitated BaSO<sub>4</sub> is suction-filtered and washed. The filtrate, together with the washings, is dried under vacuum on a water bath. The dry residue is recrystallized from half its weight of water. Electrolytic efficiency: 0.2 g./amp.-hour.



Fig. 188. Preparation of hydroxylammonium chloride by electrolysis.

II.  $HON(SO_3K)_2 + 2H_2O = (NH_3OH)HSO_4 + K_2SO_4$ 269.4 36.0 131.1 174.3

Potassium hydroxylamine disulfonate is dissolved in the smallest possible quantity of boiling water; the solution is boiled for several hours and then evaporated to induce crystallization. The solution is separated from the potassium sulfate formed; hydroxylammonium sulfate is obtained by further evaporation. The pure product is obtained by further fractional crystallization. The crude salt mixture obtained by evaporation can also be extracted with alcohol in a Soxhlet apparatus, since the hydroxylamine salt is soluble in alcohol. The chloride is obtained from the sulfate with BaCl<sub>2</sub>, as above.

### PROPERTIES:

Formula weight 69.50. Colorless crystals. Decomposes to  $NH_4Cl$ ,  $N_2O$ ,  $H_2O$  and HCl upon dry heating. Hydrolyzes, giving an acid reaction. Strong reducing agent. Oxidizes  $Fe(OH)_2$ .

### **REFERENCES:**

- F. Tafel, Z. anorg. allg. Chem. <u>31</u>, 322 (1902).
- E. P. Schoch and Pritchet, J. Amer. Chem Soc. 38, 2042 (1916).
- F. Raschig, Lieb. Ann. Chem. <u>241</u>, 183 (1887).

# Hydroxylammonium Salts

### HYDROXYLAMMONIUM PHOSPHATE

I.

# $\begin{array}{r} Na_{3}PO_{4} + 3 (NH_{3}OH)Cl = (NH_{3}OH)_{3}PO_{4} + 3 NaCl \\ (12 H_{2}O) \\ 380.1 & 208.5 & 197.1 & 175.3 \end{array}$

A solution of 500 g.  $Na_3PO_4 \cdot 12H_2O$  is prepared in one liter of hot water. The filtered solution is added to a filtered solution of 273 g. of (NH<sub>3</sub>OH)Cl in 600 ml. of hot water. The mixture is allowed to cool and is suction-filtered. Approximately 200 g. of hydroxylammonium phosphate is obtained. Further hydroxylammonium phosphate can be isolated from the mother liquor by evaporation, giving a total yield of about 90%.

II. 3 (NH <sub>3</sub> OH)Cl	$+ H_3PO_4$	+ 3 NaOH	$= (NH_3OH)_3PO_4$	+ 3 NaCl	$+ 3 H_2O$
208.5	98.0	120.0	197.1	175.3	54.0

A solution of 210 g. of (NH<sub>2</sub>OH)Cl in 200 ml. of H<sub>2</sub>O is prepared by heating. It is filtered and put into a 1.5-liter beaker placed in an empty cooling bath. A mechanical stirrer and a thermometer are placed in the solution and 115 g. of 85% phosphoric acid is rapidly added. A cooled solution of 120 g. of NaOH in 400 ml. of H<sub>2</sub>O is then added from a dropping funnel as fast as individual drops can be maintained. The final temperature should be 70°C. If the temperature threatens to exceed 75°C, the cooling bath should be filled with ice and water. The separation of hydroxylammonium phosphate usually starts toward the end of the addition of NaOH. Upon completion of the addition, the bath is filled with ice and the reaction mixture cooled to 15°C. The crystals are separated from the mother liquor on a large fritted glass filter and washed three times with 100 ml. of water. The suction is interrupted during each water addition and the crystals are mixed well with the washing water. The crystals are dried overnight in air and then under vacuum over  $P_2O_5$ . The yield is 175-180 g. of 95% hydroxylammonium phosphate. For purification the entire mass is dissolved in 1750 ml. of water at 80-85°C. Recrystallization takes place upon cooling. After filtration, washing and drying, as described above, the yield is about 140-145 g, with a purity of about 98%.

PROPERTIES:

White salt. Free hydroxylamine forms upon heating.

REFERENCES;

- R. Uhlenhut, Lieb. Ann. Chem. <u>311</u>, 117 (1900).
- P. F. Tryon, in L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-London 1950, p. 82.

#### 8. NITROGEN

### HYDROXYLAMMONIUM ARSENATE

Can be prepared in the same manner as hydroxylammonium phosphate by substituting a solution of 312 g. of  $\text{Na}_2\text{HAsO}_4 \cdot 7\text{H}_2\text{O}$  in 300 ml. of H<sub>2</sub>O for the phosphoric acid. For neutralization a solution of 40 g. of NaOH in 100 ml. of H<sub>2</sub>O (efficient cooling not necessary) is required. The crystals are washed three times with 150 ml. of water. The yield is 200-210 g. of 95% salt. After purification by recrystallization from an eightfold volume of water, the yield of 98% pure salt is about 145 g. Formula weight 241.03.

### HYDROXYLAMMONIUM OXALATE

Can also be obtained from the stoichiometric quantities of hydroxylammonium chloride, oxalic acid and sodium hydroxide (sufficient for neutralization). A diluted solution of the hydroxyl-ammonium salt (140 g. in 400 ml. of water) is used. All the solid oxalic acid is added at once with vigorous mechanical agitation, and the mixture is neutralized with a solution of 80 g. of NaOH in 200 ml. of H<sub>2</sub>O. This solution is added in the fastest possible stream of separate drops. The mixture is agitated for 30 minutes, cooled to  $15^{\circ}$ C, filtered and washed three times with 100 ml. of H<sub>2</sub>O. Since the salt at this point often contains up to 10% of oxalic acid, it must be recrystallized from eight times its weight of water. The yield is 95 g.; the purity is 99%. Formula weight 156.10.

# Hydroxylamine

### $NH_2OH$

I.

### $(NH_{3}OH)_{3}PO_{4} = H_{3}PO_{4} + 3 NH_{2}OH$ 197.1 98.0 99.1

Tertiary hydroxylammonium phosphate (20 g.) is placed in a 100-ml. distillation flask. Coarse salt, such as that obtained by seeding, is preferably used. The full scale of the thermometer inserted in the neck of the flask should be above the stopper (stem correction thermometer); the bulb reaches almost to the bottom of the flask. The flask is connected to an aspirator via a receiver. An in-line manometer is essential. The apparatus is evacuated to 13 mm. and heated carefully with an open flame. Some salt dust is readily carried over in the beginning, and this results in contamination of the product. If the receiver is cooled, the product immediately solidifies, and characteristic, pointed crystals form. The bulk of the material distills over at 13 mm. at a thermometer reading of  $135-137^{\circ}C$ . The flask is heated intermittently until the

temperature reaches 150 to  $170^{\circ}$ C; the pressure should not exceed 30-40 mm. Should the pressure exceed 40 mm., cooling must be immediately applied, lest an explosion occur (goggles must be worn!). The product distilling over at 150°C is no longer completely anhydrous. A total of 5.6 g. of distillate is obtained from 20 g. of phosphate.

The hydroxylamine obtained by distillation, which still contains water, can be purified by recrystallization from absolute alcohol. A solution of 5 g. in 100 ml. of alcohol is prepared and cooled to  $-18^{\circ}$ C. It is rapidly suction-filtered and dried in a vacuum desiccator over H<sub>2</sub>SO<sub>4</sub>, but as briefly as possible because of the volatility of the compound.

# II. CONCENTRATION BY FREEZING FROM ALCOHOLIC HYDROX YLAMINE SOLUTION

A cylindrical vessel is placed inside a Bruehl distillation receiver and secured in place with a cork stopper. The lower end of the condenser adapter is closed with a CaCl<sub>2</sub> tube. A rubber stopper with two holes, each of which carries a dropping funnel, is set in the upper side tube. The shaft of a powerful threepropeller stirrer is introduced through the center tube. The cylindrical vessel contains 34.7 g. of very finely pulverized (NH<sub>3</sub>OH)Cl, to which is added 50 ml. of absolute ethanol mixed with 0.01 g. of solid phenolphthalein. One of the dropping funnels is filled with hydroxylammonium chloride solution, identical with that found in the cylindrical vessel. The other is filled with sodium ethoxide solution (made by dissolving 11.5 g, of Na in 200 ml, of absolute alcohol). After about ten minutes of agitation, the ethoxide solution is slowly added in drops. The reaction is terminated after 1.5 to 2 hours. If the solution is reddish, some hydroxylamine salt solution from the first funnel is added until the color disappears. The solution is rapidly suction-filtered from the NaCl formed, which is rinsed with a small amount of absolute alcohol, and the filtrate plus the washing is placed in an ice-salt freezing mixture to crystallize. The crystals are suction-filtered at -18°C on a fritted glass filter. The yield is about 40%. The NH<sub>2</sub>OH still present in the NaCl residue and in the alcohol can be readily recovered. There are no losses due to decomposition. Butvl alcohol and Na butoxide can be used instead of ethyl alcohol.

### PROPERTIES:

Formula weight 33.03. M.p. 33°C. Very hygroscopic. Odorless. Produces blisters on contact with skin. Deliquescent in air. Burns on a Pt sheet when heated. Adropheated in a test tube will detonate with a very loud noise. Very volatile. **REFERENCES:** 

- R. Uhlenhut, Lieb. Ann. Chem. <u>311</u>, 117 (1900).
- E. Ebler and E. Schott, J. prakt. Chem. (2) 78, 318 (1908).
- L. F. Audrieth, Z. angew. Chem. 45, 386 (1932).
- H. Lecher and J. Hofmann, Ber. dtsch. chem. Ges. 55, 915 (1922).
- C. de Witt-Hurd and H. J. Brownstein, J. Amer. Chem. Soc. 47, 67 (1925).

# Potassium Hydroxylamine Disulfonate

# HON(SO<sub>3</sub>K)<sub>2</sub>

 $\begin{array}{l} N_{a}NO_{2} + 2 N_{a}HSO_{3} = HON(SO_{3}Na)_{2} + N_{a}OH \\ \hline 69.0 & 208.1 & 237.1 & 40.0 \end{array}$ 

A solution of 69 g. of NaNO<sub>2</sub> in the minimal quantity of water is prepared and mixed with 200 g. of ice. A cooled solution of NaHSO<sub>3</sub>, prepared either by dissolving 208 g. in the minimal quantity of water or by dissolving the same quantity in 600 ml. of technical (35%) bisulfite solution, is added in portions with agitation. Ice should still be present at the end of the addition. A cold, saturated solution of 150 g. of KCl is then added and the reaction mixture is left to stand for 24 hours. Half of the theoretical yield of disulfonate separates as compact crystals together with fine needles of trisulfonate. The nitrilotrisulfonate can be easily removed by washing and decantation. The HON(SO<sub>3</sub>K)<sub>2</sub> is then recrystallized from a small amount of hot water to which some KOH or ammonia is added (a few seed crystals are retained for this purpose from the first crystallization because the salt tends to form supersaturated solutions).

In another procedure, one mole of  $KNO_2$  and 1.1-1.2 moles of potassium acetate are dissolved in 200 ml. of ice water; 1500 g. of finely shaved ice is added and  $SO_2$  is bubbled through the agitated reaction mixture until the solution gives the characteristic odor of the gas. The temperature may not exceed 0°C. The disulfonate formed may be washed with a small amount of water and recrystallized as above.

**PROPERTIES:** 

Formula weight 269.35. Hard, shiny crystals, which readily decomposed to  $K_2SO_4$  and hydroxylamine on boiling with water.

**REFERENCES:** 

- F. Raschig, Lieb. Ann. Chem. 241, 183 (1887).
- G. K. Rollefson and C. F. Oldershaw, J. Amer. Chem. Soc. 54, 977 (1932).

# Potassium Dinitrososulfite

# $K_2SO_3 \cdot (NO)_2$

 $\begin{array}{rcr} K_2 SO_3 \ + \ 2 \ NO \ = \ K_2 SO_3 (NO)_2 \\ 158.3 & 60.0 & 218.3 \end{array}$ 

One quarter of a weighed quantity of KOH is dissolved in a small amount of water in a large flask, and the solution is saturated with  $SO_2$ . The remaining KOH and sufficient water to dissolve it are added. The solution is then diluted with an equal volume of water. A fast stream of NO from a gasometer is bubbled in until all the air is displaced. The outlet tube is then closed and more NO is introduced under the pressure of the gas holder. During this operation, the flask is cooled in a cold water bath and repeatedly shaken. Shiny crystals of  $K_2SO_3(NO)_2$  soon separate in profusion. They are rapidly suction-filtered on a fritted glass filter and briefly washed twice with ice water.

**PROPERTIES:** 

Formula weight 218.28. Shiny needles. Stable in a dry environment. Explodes upon heating. Air moisture hydrolyzes it to  $K_2SO_4$  and  $N_2O$ . Readily soluble in water but the solution soon decomposes, yielding  $N_2O$  and  $K_2SO_4$ . Reduction yields hydrazine disulfonate.

**REFERENCES:** 

- E. Weitz and F. Achterberg, Ber. dtsch. chem. Ges. <u>66</u>, 1718, 1728 (1933).
- G. A. Jeffrey and H. P. Stadler, J. Chem. Soc. (London) 1951, 1467.
- E. Degener and F. Seel, Z. anorg. allg. Chem. 285, 129 (1956).

# Potassium Nitrosodisulfonate

### ON(SO<sub>3</sub>K)<sub>2</sub>

I.

 $6 \operatorname{HON}(SO_3K)_2 + 2 \operatorname{KMnO_4}_{1616.1} = 316.1$ 

 $3 [ON(SO_{3}K)_{2}]_{2} + 2 MnO_{2} + 2 KOH + 2 H_{2}O$ 1610.0 173.9 112.2 36.0

A 5N solution of NaNO<sub>2</sub> (35 g. in 100 ml. of  $H_2O$ ) is mixed with 200 g. of ice, and 100 ml. of 5N bisulfite solution is added with vigorous stirring. Then 20 ml. of glacial acetic acid is added. After a few seconds the solution is tested to determine whether

it turns blue with a drop of KI colution. Then 25 ml. of 10N ammonia solution is added, followed by (with stirring) 400 ml. of prepared 1N permanganate solution (12.6 g./400 ml.). A thick sludge of MnO<sub>2</sub> separates; after a short settling period this is filtered off through a large fluted filter. The first 50 ml. of filtrate is passed again through the filter. A 250-ml, quantity of the darkviolet filtrate is mixed with 500 ml. of cold, saturated KCl solution. A crystalline mass of vellow needles soon separates; the mass is rapidly suction-filtered and washed briefly with ice water. II. First, 84 g. of NaHCO<sub>3</sub> and 70 g. of NaNO<sub>2</sub> are dissolved in 500 ml. of water, and then 1000 g. of ice added. Sulfur dioxide is introduced at  $-2^{\circ}C$  with stirring and cooling until the mixture is acid (pH 2). After 40 minutes, the solution turns orange-brown. Stirring is continued for another ten minutes while the introduction of SO<sub>2</sub> is discontinued; the solution turns colorless. Purified air is then bubbled through the solution for five minutes. After addition of 125 ml. of saturated Na<sub>2</sub>CO<sub>3</sub> solution, the mixture should have a pH of 9. The mixture is stirred for 1.5 hours at room temperature, 1000 ml, of distilled water and 360 g, of PbO, are added, and stirring on a water bath is continued for another 30 minutes while the temperature is maintained at 20 to 40°C. After 25 minutes of agitation, the violet solution is allowed to settle for five minutes and then filtered. The pH of the filtrate is 10. Carbon dioxide is then introduced until the pH is 7 to precipitate the Pb as the carbonate. The lead carbonate is filtered off and 30 ml. of saturated Na<sub>2</sub>CO<sub>2</sub> solution is added; after this addition the pH should be 9. Then 500 g. of KNO<sub>3</sub> is added with stirring (pH 8.5). Beautiful crystals of Frémys salt are obtained from the solution. They are very stable after drying.

SYNONYM:

Frémys salt.

PROPERTIES:

Yellow needles, yielding a violet solution in water. Monomolecular (formula weight 268.24) in solution, dimerized (formula weight 536.68) as a solid. Recrystallizable from 1N KOH. Dry, pure preparations that have been washed with pure methanol and ether are stable practically idefinitely if kept in a clean glass container under vacuum. The containers described in Part I, p. 75, are recommended (G. Brauer, private communication).

**REFERENCES:** 

H. I. Teuber and G. Jellinek, Ber. dtsch. chem. Ges 85, 95 (1952).

F. Raschig, Schwefel- und Stickstoffstudien [Studies on Sulfur and Nitrogen], Leipzig-Berlin 1924, p. 148. G. Harvey and R. G. W. Hollingshead, Chem. and Ind. 1953, 249.
D. J. Cram and R. A. Reeves, J. Amer. Chem. Soc. <u>80</u>, 3094 (1958).

# Potassium Nitrilosulfonate

 $N(SO_3K)_3 \cdot 2 H_2O$ 

 $\begin{array}{r} KNO_2 \,+\, 4\, KHSO_3 \,=\, N(SO_3K)_3 \cdot 2\, H_2O \,+\, K_2SO_3 \\ 85.1 \quad 480.7 \quad 407.5 \quad 158.3 \end{array}$ 

A solution of 75 g. of KOH in 150 ml. of water is saturated with  $SO_2$ . A solution of 25 g. of  $KNO_2$  in 100 ml. of water is added to the above hot solution with vigorous stirring. The liquid soon becomes cloudy because of precipitation of fine crystalline needles. The suspension is allowed to stand for one hour and the separated precipitate is redissolved by addition of 1500 ml. of hot water and heating. The solution must be alkaline at all times, and if necessary, a few more KOH pellets should be added. The solution is allowed to cool; the precipitate is suction-filtered, washed thoroughly with ice water, then with alcohol and ether, and finally dried in a desiccator. The yield of the dried, washed salt is 74 g. (62% of theoretical). In acid solution, disulfonate is rapidly formed. This is the reason why all bisulfite must be removed by careful washing.

### PROPERTIES:

Shiny, rhombic crystalline needles. The pure salt may be stored in a desiccator for many weeks. Slightly soluble in water and liquid  $NH_3$ . Saponified rapidly to imidosulfonate in an acid medium. Saponified to amidosulfonate at boiling temperatures.

### **REFERENCES:**

A. Claus and S. Koch, Lieb. Ann. Chem. <u>152</u>, 336 (1869).
H. Sisler and L. F. Audrieth, J. Amer. Chem. Soc. <u>60</u>, 1947 (1938).

# Potassium Imidosulfonate

### HN(SO<sub>3</sub>K)<sub>2</sub>

 $\begin{array}{rrr} N(SO_3K)_3 \cdot 2\,H_2O \ = \ HN(SO_3K)_2 \ + \ KHSO_4 \ + \ H_2O \\ 407.5 \ & 253.3 \ & 136.2 \ & 18.0 \end{array}$ 

Potassium nitrilosulfonate (38 g.) is wetted with 16 ml. of 2% H<sub>2</sub>SO<sub>4</sub> and left to stand for 24 hours. The paste is suction-filtered

and washed with 60 ml, of ice water. It is then recrystallized from a mixture of 60 ml. of water and 10 ml of concentrated ammonia. The crystals that precipitate upon cooling of the solution are washed with ice water, alcohol and ether and dried in a desiccator over H<sub>2</sub>SO<sub>4</sub>. The yield is 12 g. or 51% of theoretical.

### PROPERTIES:

Formula weight 253.3. Granular crystal aggregates or shiny plates. Very slightly soluble in cold water (but more soluble than the nitrilosulfonate). The dry salt is stable. Saponifies to amidosulfonate in boiling water.

REFERENCES:

F. Raschig, Lieb. Ann. Chem. 241, 171 (1887). H. Sisler and L. F. Audrieth, J. Amer. Chem. Soc. 60, 1947 (1938). G. J. Doyle and N. Davidson, J. Amer. Chem. Soc. 71, 3491 (1949).

# Potassium Amidosulfonate

### H<sub>2</sub>NSO<sub>3</sub>K

$$\begin{array}{rl} N(SO_3K)_3 \cdot 2 \ H_2O \ = \ H_2NSO_3K \ + \ 2 \ KHSO_4 \\ 407.5 \ & 135.2 \ & 272.3 \end{array}$$

Potassium nitrilosulfonate (60 g.) is boiled for 75 minutes with 300 ml. of water. The solution is then neutralized with 20 g. of K<sub>2</sub>CO<sub>3</sub> and evaporated to dryness. The residue is extracted with 80% alcohol for 46 hours in a Soxhlet apparatus. Cooling of the alcoholic solution yields 13.5 g. of amidosulfonate (67% of theoretical).

PROPERTIES:

Colorless crystals. Very readily soluble in water, slightly soluble in alcohol.

REFERENCES:

- F. Raschig, Lieb. Ann Chem. <u>241</u>, 176 (1887). H. Sisler and L. F. Audrieth, J. Amer. Chem. Soc. <u>60</u>, 1947 (1938).
- S. H. Maron and A. R. Berens, J. Amer. Chem. Soc. 72, 3571 (1950).

# Potassium Chloroimidosulfonate ClN(SO<sub>3</sub>K)<sub>2</sub>

 $\frac{\text{HN}(\text{SO}_3\text{K})_2 + \text{NaOCl} = \text{ClN}(\text{SO}_3\text{K})_2 + \text{NaOH}}{253.3 \quad 74.5 \quad 287.8 \quad 40.0}$ 

With efficient cooling, 7.1 g. of  $Cl_2$  is introduced into an ice cold solution of 8 g. of NaOH in 20 ml. of water (thermometer in the flask). Then 25 g. of potassium imidosulfonate is added to the solution. The mixture is heated on a water bath with occasional shaking until all solids are dissolved. It is left to stand on the water bath for several hours, then cooled to separate the  $ClN(SO_3K)_2$ .

PROPERTIES:

Water clear crystals. Unstable, decomposes to  $NCl_3$  and  $N(SO_3K)_3$ .

REFERENCE:

F. Raschig, Z. anorg. allg. Chem. 147, 1 (1925).

### Amidosulfonic acid

### H<sub>2</sub>NSO<sub>3</sub>H

 $\begin{array}{rrr} \text{CO}(\text{NH}_2)_2 \ + \ \text{H}_2\text{S}_2\text{O}_7 \ = \ 2 \ \text{H}_2\text{NSO}_3\text{H} \ + \ \text{CO}_2 \\ 60.1 \ 178.1 \ 194.2 \ 44.0 \end{array}$ 

Urea (100 g.) is added over a period of 45 minutes to 560 g. of 100% H<sub>2</sub>SO<sub>4</sub>. The mixture must be thoroughly agitated with a mechanical stirrer and efficiently cooled so that the temperature does not exceed 40°C. Then 309 g. of oleum (65% SO<sub>3</sub>) is added and the reaction mixture is left to stand for 16 hours at 42 to 45°C. The crystals are suction-filtered on a fritted glass filter and washed, first with concentrated and then with 50% H<sub>2</sub>SO<sub>4</sub>, and finally with cold methanol. The yield is 90%.

**PROPERTIES:** 

Formula weight 97.10. Colorless crystals. Very stable. Can be used as a primary titration standard in alkalimetry.

**REFERENCE:** 

P. Baumgarten, Ber. dtsch. chem. Ges. 69, 1929 (1936).

# Potassium Hydrazinedisulfonate $\label{eq:H2N2} H_2N_2(SO_3K)_2$

 $\begin{array}{rrr} N_2H_6SO_4 \ + \ 2 \ HSO_3Cl \ = \ H_2N_2(SO_3H)_2 \ + \ 2 \ HCl \ + \ H_2SO_4 \\ 130.1 \ & 233.1 \ & 192.2 \ & 72.9 \ & 98.1 \end{array}$ 

A suspension of 130 g. of dry hydrazine sulfate in 600 g. of dry pyridine is prepared. Chlorosulfonic acid (300 g.) is added in portions from an addition funnel with stirring and cooling. Finally, the mixture is heated to 90-100°C on a water bath. It is then allowed to cool, and alcohol is added with shaking. The crystalline mass that separates is suction-filtered. For purification the pyridine salt so obtained is redissolved in a small amount of water, some alcohol is added, and the mixture is evaporated in a desiccator until copious crystallization is induced. The crystals are then suction-filtered. This pyridine salt, in the form of needles with a silken sheen, can be used for the preparation of potassium azodisulfonate. The compound is readily soluble in water and very slightly soluble in alcohol. For the preparation of the potassium hydrazinedisulfonate, the crystals need not be isolated. Thus, instead of alcohol, 300 g, of ice and as much potassium hydroxide solution (d 1.39) as needed to make the solution barely alkaline are added to the impure mixture after cooling. The precipitated salt is suction-filtered and washed with cold water. The yield is about 200 g.

The ammonium salt can be prepared in the same way, by neutralizing with ammonia instead of KOH. It must be precipitated from the solution with alcohol because it is very soluble in water. It can also be readily prepared from the pyridine salt, which is then dissolved in the minimum amount of water and neutralized with ammonia. Crystallization is effected by addition of alcohol. The yield is 90%.

### PROPERTIES:

The potassium salt is very slightly soluble in water (formula weight 268.36). The ammonium salt is readily soluble in water, very slightly soluble in alcohol. Shiny plates. Pyridine salt: needles with a silken sheen, very soluble in water.

### **REFERENCES:**

- F. Raschig, Schwefel- und Stickstoffstudien [Studies on Sulfur and Nitrogen], Leipzig-Berlin 1924, p. 199.
- E. Konrad and R. Pellens, Ber. dtsch. chem. Ges. <u>59</u>, 135 (1926).

### $N_2(SO_3K)_2$

 $\frac{H_2N_2(SO_3H \cdot Pyr)_2}{350.4} + \frac{N_3OC1}{74,5} + \frac{2 \text{ KC1}}{149.1}$ 

 $= N_{2}(SO_{3}K)_{2} + N_{a}Cl + 2 HCl + 2 Pyr + H_{2}O$   $\frac{266.4}{58.5} + \frac{12}{72.9} + \frac{12}{158.2} + \frac{12}{18.0}$ 

Five grams (0.014 mole) of the pyridine salt of hydrazinesulfonic acid, described in the preceding preparation, is dispersed in a small amount of water and congealed to a slurry in a freezing mixture at  $-20^{\circ}$ C. Similarly, 0.028 mole of NaOCI (oxidation value 7%; i.e., 2g. of Cl<sub>2</sub> is introduced into an ice-cold solution of 2.3 g. of NaOH in 25 ml. of H<sub>2</sub> O) is slurried, together with 10 ml. of 12% NaOH. The hypochlorite slurry is added to the pyridine salt slurry with stirring and cooling, and after ten minutes it is allowed to warm to a maximum of  $-7^{\circ}$ C. No gas should evolve during the process. Then 10 ml. of cold, saturated KCl solution is added. The potassium azodisulfonic acid separates immediately. It is immediately placed on a well-cooled suction filter, washed with cold alcohol and cold ether, and then placed in a vacuum desiccator while still moist with ether (danger of detonation!). The material is dried over P<sub>2</sub>O<sub>5</sub>. The yield is 80-90%.

**PROPERTIES:** 

Small, yellow crystalline needles.

**REFERENCE**;

E. Konrad and R. Pellens, Ber. dtsch. chem. Ges. <u>59</u>, 135 (1926).

### Hydroxylamineisomonosulfonic Acid

### $NH_3SO_4$

 $\frac{NH_{2}OH \cdot HCl}{69.5} + \frac{ClsO_{3}H}{116.5} = \frac{NH_{3}SO_{4}}{113.1} + \frac{2}{72.9}$ 

One mole of  $NH_2OH \cdot HCl$  is placed in a 600-ml. beaker and 1.5 moles of  $ClSO_3H$  is slowly poured over it. The mixture effervesces and forms intense hydrogen chloride fumes. The mixture is allowed to cool and the salt dissolves. When the reaction begins to subside, the mixture is heated for 10-15 minutes at 70 to 80°C. However,

the temperature should not exceed 80° C since decomposition occurs at that temperature. The viscous lumps that form must be broken up by stirring. The mixture is cooled in a desiccator over NaOH. The beaker is then immersed in ice and absolute ether, cooled to  $0^{\circ}$ C, is poured over the substance. The ether reacts vigorously with the excess CISO<sub>3</sub>H. After comminution with a stirring rod and standing one hour under ether, the reaction product appears as a finely dispersed precipitate, which is suction-filtered and washed with two liters of absolute ether. The powder of hydroxylaminesulfonic acid is stored in a vacuum desiccator over P<sub>2</sub>O<sub>5</sub> because of its great hygroscopicity. The yield is almost quantitative. The purity is about 98%. It may be analyzed by iodometric titration. The acid may be further purified and thus obtained as fine crystals by dissolving the crude material in ice-cold absolute methanol and adding the filtered solution in drops to twice its volume of dry chloroform.

### PROPERTIES:

White, hygroscopic powder or water-clear prisms. Forms hydrazine with ammonia. Hydrolyzes slowly in water. Decomposes in warm, aqueous alkaline solutions to  $H_2SO_4$  and short-lived imine (NH); thus, it may serve as an amination agent. M.p. about  $210^{\circ}C$ . Has oxidizing properties.

**REFERENCES:** 

I.

- F. Sommer, O. F. Schulz and M. Nassau, Z. anorg. allg. Chem. <u>147</u>, 144 (1925).
- F. Sommer and H. G. Templin, Ber. dtsch. chem. Ges. <u>47</u>, 1221 (1914).
- G. Arens, Private communications, unpublished.
- U. Wannagat and R. Pfeiffenschneider, Z. anorg. allg. Chem. 297, 151 (1958).

# Nitrosyl Chloride

### NOCl

HSO₃ONO	+ NaCl =	= NOCl	+ NaHSO₄
127.1	58.5	65.5	120.1

A mixture of 127 g. of nitrosyl hydrogen sulfate (see p. 406) and 58.5 g. of well-dried sodium chloride is prepared in a half-liter flask equipped with a ground glass joint and a gas outlet tube. The flask is warmed on a water bath and the gases are led into a well-cooled trap. The nitrosyl chloride may also be prepared directly, thus avoiding the preparation of nitrosyl hydrogen sulfate. To accomplish this,  $N_2C_3$ , obtained from  $As_2O_3$  and  $HNO_3$ , is dried over CaCl, and then added to concentrated H<sub>2</sub>SO<sub>4</sub>. The resulting slurry is used instead of the pure nitrosyl hydrogen sulfate. It is heated after introduction of well-dried NaCl. Of course, nitrosyl hydrogen sulfate can also be prepared by addition of SO<sub>2</sub> to HNO<sub>3</sub>. The resulting thick slurry may be used directly, without preliminary isolation of the nitrosyl hydrogen sulfate. At the end of the preparation the temperature of the water bath in which the NOC1 reaction flask is immersed is preferably increased to 110°C by addition of NaCl. The NOCl condensate is purified by distillation. This is especially necessary if a non-stoichiometric quantity of pure nitrosyl hydrogen sulfate was used, but rather a slurry of impure material containing concentrated H<sub>2</sub>SO<sub>4</sub>. In this case, the NOC1 product may contain dissolved HC1.

Other preparative methods:

II.

$$KCl + 2 NO_2 = KNO_3 + NOCl$$

Liquid nitrogen dioxide is passed through a vertical, 60-cm.-long column packed with moist KCl (containing about 2.4%  $H_2O$ ), so that the total flow is approximately 20 ml. in three hours. The reaction in the column can be followed by observation of the sharp separation zones. In a properly run reaction, the reaction zone never reaches the upper end of the tube and there is no NO<sub>2</sub> in the product gas. The product gas is purified by distillation as indicated above.

III. 
$$2 \text{ NO} + \text{Cl}_2 = 2 \text{ NOCl}$$

Chlorine gas is liquefied in a suitable vessel and saturated at -50 to  $-60^{\circ}$ C with NO prepared from nitrosyl hydrogen sulfate and Hg. The crude product is passed through a glass tube heated to  $150^{\circ}$ C. It is then recondensed and purified by distillation.

PROPERTIES:

Yellowish-red gas, which condenses to a yellowish-red liquid at  $-5^{\circ}$ C. Yellowish-red crystals below  $-60^{\circ}$ C. The liquid is very corrosive to the skin; the vapor attacks stopcock grease.

### **REFERENCES:**

W. A. Tilden, J. Chem. Soc. (London) <u>27</u>, 630 (1874); <u>59</u>, 73 (1891). A. F. Scott and C. R. Johnson, J. Phys. Chem. 33, 1975 (1929).

- A. F. Scott and C. R. Jonnson, J. Phys. Chem. 33, 1975 (1929).
- M. Trautz and W. Gerwig, Z. anorg. allg. Chem. <u>134</u>, 409 (1924).

E. Wourtzel, J. Chim. Phys. 11, 243 (1913).

C. W. Whittacker, F. O. Lunström and A. R. Merz, Ind. Eng. Chem. 23, 1410 (1931).

# Nitrosyl Bromide NOBr

 $2 \text{ NO} + \text{Br}_2 = 2 \text{ NOBr} \\ 60.0 \quad 159.8 \quad 219.8$ 

Dry NO is added to 80 g. of  $Br_2$ , cooled to 0°C. During the introduction of the gas the  $Br_2$  is gradually cooled to  $-10^{\circ}$ C by placing the flask in a freezing mixture. The completely saturated liquid weighs about 28 g. more than the initial. The reaction mixture is slowly heated; the NOBr distills at 24°C. It is collected in a cooled ampoule.

PROPERTIES:

Formula weight 109.92. Dark brown liquid which readily decomposes upon warming.

**REFERENCE:** 

O. Froelich, Lieb. Ann. Chem. 224, 272 (1884).

# Nitryl Chloride

### NO<sub>2</sub>Cl

 $\begin{array}{rl} HSO_{3}Cl + HNO_{3} = NO_{2}Cl + H_{2}SO_{4} \\ 116.5 & 63.0 & 81.5 & 98 \end{array}$ 

I. A three-neck, ground glass joint flask, the center neck of which carries a mercury-seal stirrer, is charged with 63 g. of 94.4%  $\rm HNO_3$ . An addition funnel is placed in one of the side necks, with the tip of its outlet tube immersed in the  $\rm HNO_3$ , and 116.5 g. of chlorosulfonic acid is added from it. The flask should be well cooled (5 to  $10^{\circ}\rm C$ ) and vigorous agitation must be provided. The third neck of the flask carries a gas outlet tube, which leads, via a wash-flask filled with 96%  $\rm H_2SO_4$ , to a condensation trap for the  $\rm NO_2Cl$ . The yield is 73 to 74 g. (90% of theoretical).

II. Schmeisser recommends three condensation traps connected with ground glass joints. The first is filled with 122 g. of  $HSO_3Cl$ and the next with 69 g. of 100%  $HNO_3$ ; the third remains empty and is connected to a vacuum pump. Both traps are cooled to at least  $-60^{\circ}C$  and are evacuated for about five minutes. A stopcock located between the second and the third (empty) trap is then closed and cooling of the  $HNO_3$  vessel discontinued. Shortly afterward,  $HNO_3$  distills into the first trap. The level of the coolant is maintained at that of the  $HSO_3Cl$  and the  $HNO_3$  thus condenses immediately above the  $HSO_3Cl$ . When the distillation is complete, the stopcock to the third trap is opened, and the latter is cooled with liquid nitrogen. The other two traps are allowed to warm slowly to above  $-60^{\circ}C$  while still in their cooling baths. A constant stream of pure  $NO_2Cl$  evolves and condenses in the third trap. The cooling bath of the first trap may be slowly warmed to  $-20^{\circ}C$ . When the  $HNO_3$ used is incompletely anhydrous or is contaminated with nitrogen oxides, some red NOCl may form, especially toward the end of the reaction. The yield is almost quantitative.

Other preparative methods:

$$\begin{split} & \text{NOCl} + \text{O}_3 = \underset{81.5}{\text{NO}_2\text{Cl}} + \underset{32.0}{\text{O}_2} \end{split}$$
  $& \text{N}_2\text{O}_5 + \text{PCl}_5 = 2 \text{ NO}_2\text{Cl} + \text{POCl}_3 \end{split}$ 

163.0

153.4

**PROPERTIES:** 

B.p.  $-15^{\circ}$  C, m.p.  $-149^{\circ}$  C, d ( $-20^{\circ}$  C) 1.46.

208.3

108.0

**REFERENCES:** 

- K. Dachlauer (I. G. Farben), German Patent 509405 of 8/30/1929, in Chem. Zentr., <u>1930</u> II, 3832.
- H. J. Schumacher and G. Sprenger, Z. anorg. allg. Chem. <u>182</u>, 139 (1929); Z. phys. Chem. (B) <u>12</u>, 115 (1931).
- M. Schmeisser, Private communication; Z. anorg. allg. Chem. 255, 33 (1948).
- H.H. Batey and H. Sisler, J. Amer. Chem. Soc. 74, 3408 (1952).
- R. Kaplan and H. Shechter, in J.C. Bailar, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 52.
- F. Seel and I. Nógrády, Z. anorg. allg. Chem. 269, 188 (1952).

# Sodium Nitrosyl NaNO

Na + NO = NaNO23.0 30.0 53.0

A boat containing 1 g. of fresh Nais placed in a glass tube which has provision for very slow and uniform heating in a well-regulated electric furnace or a hot air jacket. A thermometer, the tip of which is placed as close as possible to the boat, is inserted in the air jacket or the electric furnace. A mixture of 18-20% NO and pure N<sub>2</sub> (free of Q<sub>2</sub>) is introduced through a glass tube set in a stopper which fits one end of the reactor tube. A wash bottle with concentrated H<sub>2</sub>SO<sub>4</sub> is attached at the inlet and outlet of the reactor tube. The gas mixture is dried with CaCl<sub>2</sub> and initially flows in at a rate of two or three bubbles per sec. When the air has been displaced, heating is begun very slowly and carefully. When the thermometer reads  $150-170^{\circ}$ C, the reaction starts, and a light-yellow mass is formed. The Na slowly swells, and a gray mass fills almost the entire cross section of the tube. The temperature is then raised to  $180^{\circ}$ C and the rate of gas flow is increased. The mass turns light-yellow to nearly white. When no further change is noted, the heating is discontinued and the substance is allowed to cool in a stream of N<sub>2</sub>. When 1 g. Na is used, the entire experiment lasts three to four hours.

Other preparative method: A product with similar composition but with somewhat different properties is obtained, according to Zintl and Harder, by the reaction of NO with a blue solution of sodium in liquid  $NH_{3}$ .

### PROPERTIES:

Soluble in  $H_2O$ , forming  $N_2O$ . Silver nitrate precipitates yellow  $Ag_2N_2O_2$  from aqueous solutions of the compound on acidification with acetic acid. Nitrite is also present in the final solution.

### **REFERENCES:**

H. Gehlen, Ber. dtsch. chem. Ges. <u>72</u>, 159 (1939).
E. Zintl and A. Harder, Ber. dtsch. chem. Ges. 66, 760 (1933).

### Sodium Nitroxylate

Na<sub>2</sub>NO<sub>2</sub>

 $\frac{NaNO_2 + Na = Na_2NO_2}{69.0 \quad 23.0 \quad 92.0}$ 

The reaction tube described on p. 87 (Fig. 69) is used for the preparation of sodium nitroxylate. Two grams of repeatedly crystallized and carefully high vacuum dried NaNO<sub>2</sub> are placed in leg b and tube c is sealed off. Fresh Na (0.4 g.) is placed in a. The ground glass cap d is put in place and its tube connected with rubber or tombac vacuum tubing to a high-vacuum apparatus (via stopcock 5). The apparatus is evacuated, stopcock 5 is closed, and
NH<sub>3</sub> is introduced via stopcock 1 and condensed on the Na in vessel 3. When 50 ml. of NH<sub>3</sub> collects in 3, stopcock 1 is closed, stopcock 5 is opened, and the apparatus is again evacuated. The stopcock in the line leading to the pump is then closed. By cooling both a and b and heating vessel 3, 20 ml. of NH<sub>a</sub> is condensed in each leg of the reaction tube. After the mixture dissolves, the Na solution is poured into the nitrite solution by tipping the tube. The glass wool placed in e prevents solid particles from reaching b. The apparatus must be cooled to a point above e to prevent boiling of the NH. The solutions must not be too concentrated. When the reaction is complete, the NH<sub>3</sub> is allowed to evaporate through valve 6. Alternately, it may be recondensed in vessel 3. If b is again cooled with liquid nitrogen, flakes of the product peel off the walls. The apparatus is then purged with pure N<sub>2</sub> dried over P<sub>2</sub>O<sub>5</sub> (copper tower, see p. 458), and as soon as it appears at valve 6, the seal at c is broken. The product can now be crushed with a long glass rod. The wash tube (Fig. 70) may be used to free the product of its nitrite impurity. The wash tube is packed with glass wool at the middle. While pure N<sub>2</sub> is introduced at the bottom, the crude product is poured in through the wide-open top. As the N<sub>2</sub> flow continues, the upper end of the tube is narrowed with a hand torch, shown in the figure. A piece of tubing is slipped over the as narrowed tube, which is then connected to the vacuum aparatus via stopcock 5. The bottom end of the wash tube is then sealed off. The tube is thoroughly evacuated and dry NH<sub>a</sub> is condensed at the bottom by cooling. After evacuation, the upper end of the tube is also sealed and the cooling device (Fig. 70) is moved into place. The tube is then carefully heated from below. Gaseous NH, flows upward through the 3-mm,-diameter insert tube into the top part of the apparatus, condenses there on the walls, and extracts the substance remaining on the glass wool filter. (Caution: safety shield and goggles must be used!) After a few hours (not more than 24), the bottom end is cooled to  $-70^{\circ}$ C, ahose is slipped over the upper end (which previously has been scratched with a glass file), the hose is connected to stopcock 5, and the end of the tube is broken off inside the hose. The ammonia is removed by suction and pure, dry N<sub>2</sub> allowed to enter. The bottom end of the wash tube is opened in the same manner; the product may now be removed in a N. stream.

PROPERTIES:

Bright yellow substance. Extremely sensitive to  $O_2$ , NO,CO<sub>2</sub>. sensitive to  $H_2O$ ; hydrolyzes to  $N_2O$ ,  $H_2N_2O_2$  and NO. Stable up to 100°C. Decomposes at 130°C, emitting sparks and evolving  $N_2$ .

**REFERENCE:** 

E. Zintl and O. Kohn, Ber. dtsch. chem. Ges. 61, 189 (1928).

# Sodium Hyponitrate

 $Na_2N_2O_3$ 

 $\begin{array}{c} (\mathrm{NH_3OH})\mathrm{Cl} + \ 3\ \mathrm{NaOC_2H_5} + \ \mathrm{C_2H_5NO_3} = \ \mathrm{Na_2N_2O_3} + \ \mathrm{NaCl} + \ 4\ \mathrm{C_2H_5OH} \\ 69.5 \qquad 204.2 \qquad 91.1 \qquad 122.0 \qquad 58.5 \qquad 184.3 \end{array}$ 

A saturated solution of 7 g. of dry hydroxylammonium chloride in warm absolute alcohol is slowly added to a concentrated solution of 7 g. of Na in absolute ethanol. The precipitated NaCl is suctionfiltered and washed briefly with absolute ethanol. The filtrates are mixed with 9 g. of ethyl nitrate. The solution is cooled and the separated Na<sub>2</sub>N<sub>2</sub>O<sub>3</sub> is suction-filtered. It is washed briefly with cold absolute ethanol and dried in a desiccator over concentrated  $H_2SO_4$ .

**PROPERTIES:** 

White salt. Readily soluble in water. Thermally unstable. The free acid has not been isolated.

**REFERENCES:** 

A. Angeli, Gazz. <u>26</u>, 18 (1896); <u>30</u>, 593 (1900).
O. Baudisch, Ber. dtsch. chem. Ges. <u>49</u>, 1181 (1916).
W.D. Bancroft, J. Phys. Chem. <u>28</u>, 1181 (1916).
L.H. Michigan and G.R. Gillette, J. Phys. Chem. <u>28</u>, 754 (1924).

## SECTION 9

# Phosphorus

### **R. KLEMENT**

### White Phosphorus

 $\mathbf{P}_4$ 

#### VERY PURE PHOSPHORUS

Commercial phosphorus is usually rather pure. It is, however, repurified for special purposes, either by double steam distillation (I) or by treating molten phosphorus with very dilute chromosulfuric acid (II).

I. Commercial phosphorus (100 g.) and 0.5 liter of water are placed in a four-liter flask connected to a steam generator, a tank of  $CO_2$ and a condenser. An adapter on the condenser outlet connects to a receiver, partially filled with water. The adapter tube should empty below the water surface. The receiver is heated to 30°C and the air in the apparatus is displaced with  $CO_2$ . Steam is then introduced into the flask at a rate sufficient to keep the contents vigorously and continuously boiling. As  $CO_2$  is slowly and continuously passed through, the phosphorus steam-distills in colorless, strongly refracting droplets which fall to the bottom of the receiver and eventually solidify. The receiver water is periodically changed. It takes about eight hours to distill 50 g. of phosphorus.

If the product is still not completely As-free, a second distillation yields pure material. To test for As, the phosphorus is oxidized with HNO<sub>3</sub>. The evaporated solution is taken up in HCl and precipitated with H<sub>2</sub>S. The precipitate is dissolved in  $(NH_{4})_2CO_3$  and the solution acidified with HCl. The arsenic sulfide which precipitates is dissolved in NH<sub>4</sub>OH and H<sub>2</sub>O<sub>2</sub>, and the As precipitates as NH<sub>4</sub>MgAsO<sub>4</sub> · 6 H<sub>2</sub>O.

II. Commercial white phosphorus is melted under dilute chromosulfuric acid, stirred vigorously with a glass rod, and after solidification, thoroughly washed with distilled water. Melting under acid must be repeated until the phosphorus is no longer yellow.

#### PROPERTIES:

Atomic weight 30.975. Pure phosphorus is white, translucent, soft and readily cut. It is brittle when cold and shows a crystalline structure at break surfaces. Beautiful crystals are obtained by

evaporation of a solution of white phosphorus in  $CS_2$  or in benzene or by slow vacuum sublimation in the absence of light. Because of its low ignition temperature (60°C), phosphorus should not contact warm objects, and thus can be cut only when wet, preferably under water. On contact with skin, phosphorus produces deep, difficultto-heal burns. Only water is used to put out a phosphorus fire on the skin. Skin burns should be treated with compresses soaked in a 1% solution of CuSO<sub>4</sub>. A physician should be called as soon as possible. A phosphorus burn must never be treated with a salve or an oil because white phosphorus is soluble in these substances and thus the burn can spread further on the skin. M.p. 44.1°C; d 1.82. Almost insoluble in water, somewhat soluble in alcohol and acetic acid; readily soluble in ether, benzene, fatty oils, PCl<sub>3</sub>, PBr<sub>3</sub>, liquid NH<sub>3</sub> and liquid SO<sub>2</sub>. Miscible in all proportions with CS<sub>2</sub>.

**REFERENCES:** 

- I. E. Nölting and W. Feuerstein. Ber. dtsch. chem. Ges. 33, 2684 (1900).
- II. I. Pakula. Brief communication.

GRANULAR WHITE PHOSPHORUS

Fine phosphorus powder is obtained by shaking pieces of phosphorus with warm water or warm urea solution (thick-wall, closed flask, mechanical shaker). As the liquid cools, the phosphorus solidifies to granules of various sizes.

REFERENCE:

A. Michaelis. Liebigs Ann. Chem. 310, 56 (1960).

### **Red Phosphorus**

Red phosphorus occurs in various forms which, according to Krebs, are not different modifications. Commercial red phosphorus is amorphous on x-ray analysis, and crystallizes exothermally above  $450^{\circ}$ C when heated at a rate of 1°/min. Depending on material history and the heating rate, the melting point varies between 580 and 610°C. The initially diffuse x-ray diffraction pattern becomes increasingly sharper on long heating to just below the melting point and ultimately becomes identical to that observed with Hittorf's phosphorus. This treatment increases the density from 2.10 (red phosphorus) to 2.32 (Hittorf's phosphorus). According to Schenck and Wolf, bright red phosphorus cannot be obtained pure, and Krebs refers to it as a "mixed polymerizate" containing varying amounts of bromine.

I. Red Phosphorus. Commercial red phosphorus is purified as follows: Portions of 100 g. are boiled for 24 hours with a 7% NaOH solution. After removal of the hydroxide, the phosphorus is boiled

for 24 hours with water, washed with cold water until the alkaline reaction disappears (a centrifuge may be used here to advantage) and dried in vacuum on clay plates over  $P_2O_5$ . Phosphorus which has stood for some time must be freshly washed before use because of its oxidation on exposure to air.

II. Hittorf's Phosphorus. Phosphorus dissolved in molten lead separates on cooling as a crystalline form (Hittorf's phosphorus). This material can be readily prepared via the Stock and Gomolka procedure, as improved by Pakulla.

White phosphorus, purified with very dilute cleaning solution, is weighed under ice water, quickly and carefully dried with filter paper and acetone, and placed at the bottom of a Pyrex ampoule which has been purged with CO<sub>2</sub>. The ampoule, 6-10 cm. long, has an inner diameter of 10-12 mm. and a wall thickness of 1.5-2 mm. The free space above the phosphorus is filled as completely as possible with strips of very pure, Zn-free lead. The ampoule should be filled up to the constriction. For example, 0.7 g. of phosphorus and 20 g. of lead or 1.1 g. of phosphorus and 35 g. of lead are used. Larger charges yield poorly formed crystals. The ampoule is evacuated with an aspirator, sealed by fusing, and placed horizontally in an electric furnace capable of delivering a large amount of heat. If the ampoule is made of Vycor and has been carefully sealed, no protective iron tube is needed, provided the temperature does not rise above 640°C. The ampoule is heated to 625-640°C (the thermocouple should be located next to it). The temperature of the furnace is gradually reduced to about 400°C over a period of several days. The heat is then shut off. After complete cooling, the ampoule is chilled in a salt-ice mixture so that the small amount of white phosphorus present does not burst into flame on opening. The ampoule contains mostly a small amount of crystalline Hittorf's phosphorus in the form of fine needles. The bulk of the phosphorus is occluded in the lead. It is recovered by electrolytic dissolution of the lead used as an anode.

The apparatus shown in Fig. 189 is used for this purpose. The vessel, which is made by breaking off the top of a one-liter bottle, contains the electrolyte, which is prepared from 200 g. of lead acetate and 800 ml. of 6% acetic acid. A lead plate A, 2 mm. thick, serves as the cathode. Soldered to it is a copper lead wire, which is insulated from the electrolyte by means of a rubber sheath and a glass tube. The Pb bar B, which serves as the anode, has a hole bored through its rounded end and is suspended from the Cu lead wire by a platinum wire sling so that one third of it is immersed in the liquid. A watch glass C, supported by glass rods halfway down through the liquid, serves as a trap for the P. The anode is surrounded by a glass cylinder D, 5 cm. in diameter, which protrudes 1 cm. into the fluid (the supporting clamp is not shown). The lead bar is then gradually lowered without letting the platinum wire come in contact with the solution. The current is adjusted to 0.5 amp.



Fig. 189. Isolation of Hittorf's phosphorus. A-Lead cathode; B-lead rod containing Hittorf's phosphorus; C-watch glass; D-glass cylinder.

The lead is deposited on the cathode in well-formed crystals, which are pressed together from time to time with a glass spatula. The phosphorus collects on the watch glass as a reddish-brown crystalline powder. After electrolysis, the PbO<sub>2</sub> present is removed by suspending in water. The powder is then refluxed for 24 hours with 20% HCl in a CO<sub>2</sub> atmosphere. The acid is renewed three or four times until no further lead dissolves. The residue is centrifuged, washed with cold H<sub>2</sub>O, and dried in vacuum over  $P_2O_5$ . In some cases, glass splinters must be removed with hydrogen fluoride.

#### PROPERTIES:

Hittorf's phosphorus forms tetragonal plates which are some-

times a few millimeters long and 0.5 mm. thick. Depending on layer thickness, they range from bright to dark red. The sublimed needles are bright red to dark reddish-brown. Hittorf's phosphorus is never completely free of lead, the lead impurity ranging from 1.5 to 3%. A sample containing 1.5% lead has a density of 2.36. By extrapolation, the density of pure Hittorf's phosphorus is computed to be 2.31-2.33. The term "metallic" P for this modification is considered incorrect by Stock and Stamm because the material is not electrically conductive.

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III. Bright-red P. Bright-red P is obtained in Wolf's procedure by reduction of  $PBr_3$  with Hg.

A mixture of 55 g. of Hg and 51 g. of PBr<sub>3</sub> (both as pure and dry as possible) is heated for two days at 100°C in a sealed tube with constant shaking, special care being taken that the Hg does not adhere to the tip of the tube. The tube is then heated for one day in a furnace at 130°C; the temperature raised to 170°C the following day. The product is carefully broken up and extracted six times with absolute ether in an extraction apparatus (two hours each time). After drying, it is heated once or twice in vacuum under dry, oxygen-free CO<sub>2</sub> to sublime the HgBr<sub>2</sub> byproduct. The residue consists of about 87% P, the remainder being Hg\_Br, and HgBr. This material is again placed in a combustion tube, a drop of PBr<sub>3</sub> is added for each gram of the substance, and after evacuation, the tube is sealed by fusing. The 0.5-m.-long tube containing the material is placed in a furnace; one third of the tube is allowed to protrude out of the furnace. The tube is carefully heated for one day at 220-240°C and is then allowed to cool. The product HgBr<sub>2</sub> readily crystallizes in the tip of the tube. It is freed of the bulk of the remaining PBr, by a six-hour extraction, the ether being changed three times, and residual HgBr, is distilled off in CO2 atmosphere at 25-30 mm. The brown-black residue becomes vermillion on cooling. If quantitative analysis shows this material to be still contaminated, it is remelted with PBr<sub>a</sub> and the procedure repeated. If the distillation is too slow, Hg<sub>2</sub>Br<sub>2</sub> is formed, and whereas the PBr<sub>3</sub> distills off, the nonvolatile Hg<sub>2</sub>Br<sub>2</sub> remains. In that case, repetition of the procedure only serves to transform the Hg(I) into Hg(II) salt.

#### **PROPERTIES:**

The bright-red phosphorus described here is quite different from that obtained by R. Schenck from white P and PBr<sub>3</sub>. Its color varies between that of red lead and vermillion and deepens (reversibly) to brown-black on heating to  $250^{\circ}$ C. According to Wolf, bright-red P is not an allotropic modification. Rather, it is to be considered a variant of the common "red" P, from which it is differentiated by its smaller particle size. Insoluble in ether and CS<sub>2</sub>. Not darkened either by liquid or aqueous NH<sub>3</sub>. The ignition temperature in air is about  $300^{\circ}$ C. In moist air it oxidizes slowly. d.  $(24^{\circ}$ C) 1.876.

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### **Black Phosphorus**

According to Bridgman, black P is formed by a pressure of 12,000 atm. acting for about one half hour on white P at  $200^{\circ}$ C.

This modification may be produced at room temperature by a shock wave of 100,000 atm. (Günther, Geselle and Rebentisch). According to Krebs, Weitz and Worms, large quantities of black P can be obtained by the catalytic action of Hg on white P at  $370^{\circ}$ C.

A mixture of 50 g of distilled, white P and 50 g of Hg is placed in an ampoule filled with pieces of copper-plated welding rods. At the same time, 0.5 g of black P, which has been well pulverized beforehand in an atmosphere of  $N_2$ , is added as seed crystals. The ampoule is fused shut and gently heated until the white P melts. It is then shaken to achieve a good mix. As a result, a layer of seed crystal powder adheres to the newly amalgamated surface of the welding rod. The ampoule is heated in a protective iron tube to 220°C and then, over a period of two days, to 370°C. After a total of eight days, black P forms quantitatively. Its surface sometimes shows traces of white and red phosphorus.

To produce the seed crystals, a small ampoule filled with freshly distilled white P and 30-40 at. % Hg is placed in a furnace preheated to about  $370^{\circ}$ C. It is left there for three days at this temperature. It is then heated for one day at  $380^{\circ}$ C, one day at  $390^{\circ}$ C and three to four days at  $410^{\circ}$ C. The well-formed spherules of black P can be easily separated from the other material.

To extract the crude product from the admixed Hg, the pulverized sample is placed next to a piece of Pb and heated in an evacuated ampoule for several days at  $300-450^{\circ}$ C. After repeating the process with the repulverized sample and fresh Pb, the remaining Hg amounts to about 1 at. %. If gold is used instead of Pb in the second amalgamation the amount of Hg after heating to between  $370^{\circ}$ C and  $440^{\circ}$ C is reduced to about 0.5 at. %. The Hg content cannot be further reduced by this or any other method.

#### PROPERTIES:

Black P (containing Hg) takes up  $O_2$  and  $H_2O$  in moist air and becomes coated with a layer of viscous fluid. After three weeks the weight gain of a pulverized sample amounts to about 13%. At higher temperatures the liquid layer is formed more rapidly. This layer protects the black P from air, and therefore it cannot be ignited with a match. Concentrated nitric acid (d. 1.4) reacts explosively with a fine pointed flame. Concentrated sulfuric acid is reduced to  $SO_2$  at about 150°C. Warm 3-6%  $H_2O_2$  reacts somewhat more rapidly with black P than with red, while bromine vapor or bromine dissolved in benzene attacks the black modification more slowly than the red. Heating for eight hours at 560°C causes transformation to red P. Forms rhombic crystals and has a layer lattice. Conducts electricity and rectifies AC current. d. 2.7-3.0. REFERENCES:

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### **Colloidal Phosphorus**

Colloidal phosphorus can be prepared, according to Svedberg (I), by pulverizing red P with an electric spark. Roginsky and Schalnikoff (II) obtained a hydrosol of P through simultaneous condensation of P vapor and water vapor at the temperature of liquid air and subsequent thawing of the substance. According to German Pat. 401,049, colloidal P can be obtained by mixing a solution of white P in  $CS_2$ with water in the presence of a protective colloid suspension agent (III).

I. A conical aluminum vessel containing red P and isobutyl alcohol is placed in a glass funnel with its upper edge ground flat (see Fig. 190). A glass cover is provided with a center hole for an Al electrode. The funnel insulates and supports the apparatus very effectively. The aluminum vessel and wire are attached to a power



Fig. 190. Preparation of colloidal phosphorus.

supply (induction coil with a capacitor across the secondary) and a spark is produced within the liquid. The aluminum wire should be grounded in order to allow adjustment with the ungloved hand. The resulting colloid is practically colorless (slightly yellow) in transmitted light and flesh-colored in reflected light.

II. Water is placed in side tubes  $A_1$  and  $A_2$  of the apparatus shown in Fig. 191. Adapter *B* contains phosphorus which can be vaporized by means of an enclosing electric furnace. Before the run, the water is frozen by means of liquid nitrogen so that the apparatus can be evacu-

ated. Liquid nitrogen is poured into vessel D and then the furnace is turned on, vaporizing the P. The ratios of solvent to P are regulated by the temperature of tubes  $A_1$ ,  $A_2$  and B. After a sufficient amount of mixture has settled on the wall of D, the liquid nitrogen is removed; the mixture is melted and flows into vessel C.

The sol thus obtained is polydisperse, probably due in part to the incompleteness of the mixing, and in part to the subsequent enlargement of the particles of the solid mixture on melting. III. One part of a 50% solution of P in CS<sub>2</sub> is shaken vigorously

III. One part of a 50% solution of P in  $CS_2$  is shaken vigorously with two parts of Turkey red oil. A milky white liquid forms,

which gives a stable emulsion on dilution with water (e.g., 2,000 parts).

After removal of the  $CS_2$ , the solution releases a vapor which phosphoresces in the dark.

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### Phosphine and Diphosphine

PH<sub>3</sub>

Phosphine (hydrogen phosphide) can be prepared by any one of the following methods:

I. action of potassium hydroxide on white phosphorus:

P<sub>2</sub>H₄

 $4P + 3KOH + 3H_2O = PH_3 + 3KH_2PO_2,$ 123.9 168.3 54.0 34.0 312.3

II. reaction of calcium phosphide with water:

 $\begin{array}{c} Ca_{3}P_{2} + 6 H_{2}O = 2 PH_{3} + 3 Ca(OH)_{2}, \\ 182.2 & 108.1 & 68.0 & 222.3 \end{array}$ 

III. treatment of phosphonium iodide with potassium hydroxide

 $PH_4I + KOH = PH_3 + KI + H_2O.$ 161.9 56.1 34.0 166.0 18.0

Pure  $PH_3$  will be obtained only by method III, the other procedures yield a product contaminated to varying degrees with diphosphine,  $P_2H_4$  and  $H_2$ . Liquid  $P_2H_4$  can be obtained at the same time.

I. PREPARATION FROM PHOSPHORUS AND POTASSIUM HYDROXIDE

A round-bottom, 3-liter flask K (see Fig. 192) is closed with a four-hole rubber stopper. The following are inserted through



Fig. 191. Preparation of colloidal phosphorus  $A_1$ ,  $A_2$ ) for water; B) for phosphorus; C) collecting tube for the colloidal phosphorus; D) cold finger.



Fig. 192. Preparation of phosphine. K) reaction flask;  $W, F_1 - F_5$  traps; Tr) drying tube.

the holes: a hydrogen inlet tube, a separatory funnel for potassium hydroxide, an outlet tube with a small reflux condenser, and a thermometer (not shown in Fig. 192). A series of vessels is attached to the outlet tube as follows: an ice-cooled trap W, two drying tubes  $\mathcal{I}r$ , filled with solid KOH, four gas traps  $F_1 - F_4$ , and two traps  $F_5$  and  $F_6$  for fractionation of the liquefied gases. Each trap is provided with a manometer for control of the fractionation (not shown in Fig. 195).

Electrolytic hydrogen is passed through traps cooled to  $-180^{\circ}$ C and filled with activated charcoal, and thereafter it is freed of O<sub>2</sub> by contact with platinized asbestos catalyst.

Flask K, about half filled with white P and dilute potassium hydroxide, is heated to +60°C after the air has been displaced with H<sub>2</sub>. The evolving gases, containing a great deal of moisture and carried along by the rapid stream of H<sub>2</sub>, are partially freed of water in the reflux condenser, the trap W, and the drying tubes Tr. The remainder of the water vapor can only be removed by fractionation at low temperature. The traps are cooled to the following temperatures:  $F_1$ : -90°C,  $F_2$ : -100°C,  $F_3$  and  $F_4$ : -180°C. Traps  $F_1$  and  $F_2$  collect mainly P<sub>2</sub>H<sub>4</sub>, while the PH<sub>3</sub> condenses in traps  $F_3$  and  $F_4$ . The PH<sub>3</sub> is carefully removed from these traps by fractional distillation rejecting everything but the lowest boiling fractions.

The contents of traps  $F_1$  and  $F_2$  are subjected to careful fractionation between  $-70^{\circ}$ C and  $-100^{\circ}$ C in order to obtain the  $P_2H_4$ . The entrained water remains as a residue. When necessary, the material is quickly distilled at  $-50^{\circ}$ C and the first cut is discarded; thus, any traces of PH<sub>a</sub> which may have formed are removed.

At the end of the run, the flask contents are allowed to cool in a stream of  $H_2$  until complete solidification of the phosphorous occurs. The apparatus may only be disassembled after the  $PH_3$  has been completely removed by the  $H_2$  stream. The P must be washed with water until free of alkali to prevent further production of  $PH_3$ .

### **II. PREPARATION FROM CALCIUM PHOSPHIDE**

The apparatus shown in Fig. 195 is used. However, flask K has a capacity of only one liter. In order to decompose any  $P_2H_4$ , as well as for drying, the product gas is passed through long tubes filled with soda-lime and  $P_2O_5$  (instead of tubes *Tr*) and is condensed in two traps cooled with liquid nitrogen.

Commercial  $Ca_3P_2$  is placed in flask K and the apparatus is completely filled with electrolytic  $H_2$  (for purification of the latter, see p. 111). By means of the separatory funnel, dilute hydrochloric acid is then added drop-by-drop. The PH<sub>3</sub>, which collects in the trap, is carefully fractionated (once) and only the lowest boiling fraction is collected.

According to Baudler and Schmidt, very pure  $P_2H_4$  (7-8 g. per run) can be obtained in the apparatus shown in Fig. 192a. To obtain larger quantities, condensates collected from several runs are combined prior to distillation.

Commercial Ca<sub>3</sub>P<sub>2</sub> (as freshly prepared as may be obtained) is crushed to pea-size grains. The material is then sieved. The feed bulb M is loaded with 375 g. of  $Ca_3P_2$ . Then all oxygen is displaced from the entire apparatus through repeated evacuation and filling with very pure nitrogen. Stopcock 2 is then closed and 500 ml. of  $O_2$ -free water is added to flask U from dropping funnel T. Flask U is placed in a 60-65°C bath and small portions of the phosphide are added to it by rotating bulb M in the joint. The addition time should be no less than three hours. To prevent the distillation of the water from U into M, joint L is cooled with a stream of compressed air. The first gas fraction liberated passes through stopcock 1 and connection a into section III. It is collected under the safety bell which is immersed in water and is connected to a burner. The gas fraction is mixed with city (or natural) gas and is burned. Other gases, liberated in various sections of the apparatus, are also collected under the bell and similarly disposed of.

Stopcocks 4 and 8 are now closed and stopcock 2 is opened. The gas then passes through the reflux condenser and the KOH tube  $Tr_1$ , where the entrained water vapor is removed. The  $P_2H_4$  (plus some water) condenses in trap A, which is cooled to  $-78^{\circ}$ C. The noncondensable gases pass through  $A_1$  and are either released to section III through stopcock 10 or, if it is desired to recover the  $PH_3$ , are condensed in the  $-196^{\circ}$ C trap G.

If the first run is to be followed by another one, flasks U and M are removed while the apparatus is flushed with a fast stream of N<sub>2</sub>. Identical fresh flasks, already charged with reagents and free of oxygen, are immediately substituted.

Before the start of the purification sequence, all remaining product gases are removed from the apparatus with a  $N_2$  stream



Fig. 192a. Preparation of diphosphine. I. Reaction and condensation section (up to stopcock 4), also used for distillation and transfer of product to storage vessels. II. section used for condensation of  $PH_3$  (connected to section I at S). III. Safety apparatus for the decomposition of phosphines. A to G are condensation traps; M, 300-ml. flask; U, 1000-ml. flask; HV, high-vacuum connection; WV, aspirator connection;  $\neg$  separate connections to section III.

and are burned in section III. Stopcock 3 is then closed. The threeway stopcocks at traps  $A_1$  and G are turned so that they communicate only with their respective pressure release valves, but are otherwise closed. Trap A is then evacuated with an aspirator through stopcock 8 in order to remove dissolved PH<sub>3</sub>. Stopcock 4 is then opened and, at a pressure of 5 mm., a small forerun is removed into the liquid-nitrogen-cooled trap E. The main  $P_2H_4$ fraction is then distilled from A (at  $-35^{\circ}$ C) through the KOH tube  $Tr_{2}$  and into B, which is cooled to -196°C. This treatment removes traces of water. Distillation of 7 g. of product takes 2-3 hours. Stopcock 4 is then closed. The temperature in B is raised to  $-60^{\circ}$ C and the P<sub>2</sub>H<sub>4</sub> is distilled at  $10^{-3}$  mm. into trap C, cooled to  $-196^{\circ}$ C. The procedure is repeated (this time from C into D). Only the middle fraction is collected in each case. Then stopcock 5 is closed and N<sub>2</sub> is introduced at  $f_{\bullet}$ . The pure P<sub>2</sub>H<sub>4</sub> is forced out through the siphon of trap 1 into a series of vessels, cooled to  $-78^{\circ}$ C, and attached at R. The inlet of each of these vessels is narrowed to a capillary which may then be fused to form a seal. The series of vessels is connected to a mercury pressure release valve, which can be disconnected from the system by a stopcock. In this way, no Hg transfers into the receivers during the high vacuum distillation.

At the end of this procedure, the system is reevacuated to 2-5 mm. with stopcock 5 initially closed. Nitrogen is then introduced until the pressure is almost equal to atmospheric. The series of receivers is then sealed off with a torch and is stored at  $-78^{\circ}$ C. However, they remain connected to the pressure release valve. All these operations must be conducted, as far as possible, in the absence of light. All glassware must be precleaned in the usual way, after which it is repeatedly rinsed with 50% ammonia solution and with distilled water.

#### III. PREPARATION FROM PHOSPHONIUM IODIDE

#### A) WITH POTASSIUM HYDROXIDE

Pea-sized pieces of  $PH_4I$  are mixed with small pieces of glass in an Erlenmeyer flask. The flask is closed with a two-hole rubber stopper. A separatory funnel and a glass outlet tube are inserted in the holes. By adding dilute potassium hydroxide (about 1:2) in drops, a steady stream of very pure  $PH_3$  is obtained without a rise in temperature. However, if the potassium hydroxide is not added slowly enough, then some  $P_2H_4$  may also form. One liter of  $PH_3$  is delivered by 7.3 g.  $PH_4I$ .

#### B) WITH WATER CONTAINING ETHER

About 10 g. of  $PH_4I$  is placed in an Erlenmeyer flask. The flask is closed with a two-hole rubber stopper. A separatory funnel and a glass outlet tube are inserted through the holes. First, ordinary ether is added to the flask from the separatory funnel. Its water content is sufficient to initiate evolution of  $PH_3$ . As soon as the moisture content of the ether decreases and the gas stream becomes slower, one or more drops of water are added from the separatory funnel, thus restarting gas evolution. This lasts for about eight hours.

A small Kipp generator may also be used for the reaction.

The PH<sub>3</sub> obtained via this method may be contaminated with ether vapor.

SYNONYM:

(Gaseous) hydrogen phosphide.

PROPERTIES (PH<sub>3</sub>):

Colorless, very poisonous gas with a peculiar, acetylenelike odor. (The odor of ordinary acetylene is due to the presence of small amounts of  $PH_{3.}$ ) Ignites in air at about  $150^{\circ}C$ , particularly when very dry. Ignites spontaneously at room temperatures only if contaminated with  $P_2H_4$  during the preparation.

M.p. -132.5°C, b.p. -87°C.

Only slightly soluble in water: at room temperature one volume of water absorbs 0.112 volume of PH<sub>3</sub>.

PROPERTIES  $(P_2H_4)$ :

Colorless liquid. The vapor ignites spontaneously in air. M.p.  $-99^{\circ}$ C, b.p.  $+51.7^{\circ}$ C. Decomposes on rough surfaces and in the presence of traces of acid (particularly in light) into PH<sub>3</sub> and amorphous, yellow "solid hydrogen phosphide," which must be considered a "mixed polymerizate."

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### Sodium Dihydrogenphosphide

### NaPH2

Sodium dihydrogenphosphide is prepared by addition of  $PH_3$  to a blue solution of Na in liquid  $NH_3$  (Royen method):

 $\begin{array}{rrr} Na + PH_3 = NaPH_2 + \frac{1}{2}H_2 \\ 23 & 34 & 56 & 1 \end{array}$ 

Approximately 25 ml. of liquid  $NH_3$ , dried over Na and fractionated at least once, is condensed on 0.5 g. of Na (sealed in a glass ampoule; see Fig. 265). After breaking the ampoule and dissolving the Na, purified  $PH_3$  is introduced in a stream of  $N_2$  into the blue solution until the blue color disappears. The addition is continued for some time. The apparatus is protected from the atmosphere by means of a pressure release valve, according to the method of Zintl, Goubeau and Dullenkopf (see p. 56). The  $NH_3$  is distilled off through this valve at the end of the reaction. After reaching room temperature, the molten diammoniate  $NaPH_2 \cdot 2NH_3$  remains behind as a yellow liquid, which loses  $NH_3$  in vacuum, forming a white salt. The yield, based on the Na used, is quantitative.

PROPERTIES:

White crystalline powder, which splits off  $PH_3$  on heating above  $60^{\circ}C$ :  $2NaPH_2 = Na_2PH + PH_3$ . At  $95^{\circ}C$ , this transformation is complete. The  $Na_2PH$  is yellow. In water,  $NaPH_2$  decomposes to  $PH_3$  and NaOH.

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#### Phosphonium Iodide

#### PH₄I

Phosphonium iodide is prepared in the Baeyer method (I) by decomposition of phosphorus iodide with water. The mechanism of the transformation is not entirely clear; the reaction proceeds roughly according to the equation:

 $\begin{array}{l} 2 P + I_2 + 4 H_2 O = P H_4 I + H I + H_3 P O_4. \\ 62.0 \quad 253.8 \quad 72.1 \quad 161.9 \quad 127.9 \quad 98.0 \end{array}$ 

After Baeyer's method of preparation is presented, that of Hofmann (II) will be given, because, according to the latter author, it is advantageous to work with large quantities.

I. A fairly large tubular retort is closed off with a stopper. An inlet tube for dry  $CO_2$  and a separatory funnel are inserted into the stopper. White P is then placed in the retort and dissolved in dry  $CS_2$ . Then 175 g. of I is added in small portions while the vessel is efficiently cooled. After the reaction is complete, the  $CS_2$  is distilled off. The last traces of  $CS_2$  are removed in a stream of dry  $CO_2$ , with gentle heating of the retort. On cooling, the condenser is replaced with a long, large diameter, thinwall glass tube. The free end of the tube is connected to a gas tube, which in turn, ends in awater-filled flask, without, however, touching the water surface. By means of a separatory funnel, 50 ml. of water is added in small portions to the phosphorus iodide. On each addition, a vigorous reaction takes place, producing HI.

The latter is absorbed by the water in the receiver, while  $PH_4I$  sublimes onto the wall of the retort and into the large glass tube. After all the water has been added, the retort is heated, at first gently and then to a dull glow, so that the  $PH_4I$  is transferred completely into the glass tube. After cooling, the tube is separated from the retort, one end is closed with a stopper, and the  $PH_4I$  adhering to the wall is removed with a long wire. The yield is about 120 g.

II. In the Hofmann method, which is useful for larger quantities, 400 g. of P is dissolved in an equal weight of CS<sub>2</sub> in a retort of at least one liter capacity. It is then reacted with 680 g. of I. The CS<sub>2</sub> is then distilled off over a period of several hours, using a water bath. The decomposition uses 240 g. of water, added in very small portions, with gentle heating and in a constant stream of dry CO<sub>2</sub>. The HI produced is trapped in water in two wash bottles connected to a glass tube 1.3 to 1.5 m. long and 3 to 4 cm. in diameter. The wash bottles are very large to prevent backup of water into the retort. Dilute hydriodic acid is added to the first flask for better absorption. To avoid explosion of the hot PH<sub>4</sub>I vapor, the apparatus should be protected from air. After the reaction with water is complete, the retort is heated, at first gently and then, at the end of the sublimation, to a dull glow. The sublimation takes 8 to 9 hours. Hydriodic acid is obtained as a byproduct. This, however, is somewhat contaminated with H<sub>3</sub>PO<sub>4</sub>.

### PROPERTIES:

Large water-clear crystals, with a diamond glitter. Tetragonal crystal system. Sublimes at room temperature. Instantaneous decomposition with water, accompanied by formation of  $PH_3$  (see p. 529) and HI. Must therefore be stored out of contact with atmospheric moisture.

B.p. 80°C.

**REFERENCES:** 

I. A. von Baeyer. Liebigs Ann. Chem. 155, 269 (1870).

II. A. W. von Hofmann. Ber. dtsch. chem. Ges. 6, 286 (1873).

# Thiophosphoryl Chloride

 $PSCl_3$ 

Thiophosphoryl chloride is prepared by addition of S to PCl<sub>3</sub>:

 $PCl_3 + S = PSCl_3$ 137.3 32.1 169.4 According to German Pat. 675,303(I), the reaction can be carried out by passing  $PCl_3$  vapor over S at  $160^{\circ}C$ . Alternately, in the method of Knotz (II) it is prepared by reacting  $PCl_3$  with S, using AlCl<sub>3</sub> as a catalyst.

### I. PREPARATION ACCORDING TO GERMAN PAT. 675, 303

An oil bath is used to heat 250 g. of S at a bath temperature of  $195^{\circ}$ C. The internal temperature of the reaction vessel is then about  $160^{\circ}$ C. In another vessel, 500 ml. of PCl<sub>3</sub> is heated to vigorous boiling. The vapors are passed over the molten S, which is vigorously stirred. The PSCl<sub>3</sub> product and the unreacted PCl<sub>3</sub> are condensed in a reflux condenser and flow back into the PCl<sub>3</sub> removed by simple fractionation. Very pure PSCl<sub>3</sub> (about 205 g.) is obtained between 118 and  $122^{\circ}$ C. The unreacted PCl<sub>3</sub> is recovered and, together with the unreacted S, can be used in a new run.

According to unpublished data of R. Klement, the following procedure can be carried out with simple equipment available in any laboratory. The S is melted in a round-bottom, ground-joint flask placed in an oil bath. The flask carries a ground-glass adapter fused to a downward-tilted condenser. The condenser is connected through an adapter to a receiver, which is protected from the air with a CaCl<sub>2</sub> tube. The ground-glass adapter on the flask is also provided with an inlet tube dipping into the molten S. This tube is attached to a Claisen flask, provided with an insert thermometer. The PCl<sub>3</sub> is brought to the boil in the Claisen flask and its vapor is passed through the molten S in a moderately fast stream of dry CO<sub>2</sub> from a steel cylinder. After boiling off the PCl<sub>2</sub> in the Claisen flask, the liquid collected in the receiver is transferred into the Claisen flask to gradually concentrate the PSCl<sub>a</sub>. By controlling the boiling temperature, it is possible to avoid passage of the PSCl<sub>3</sub> product over the S. This means a saving of time without limiting the yield. From 100 g. of S and 200 g. of PCl., about 80 g. of PSCl. can be obtained within 6 hours. This can then be purified by fractionation. One disadvantage of this procedure is that it requires constant supervision.

### II. PREPARATION BY THE KNOTZ METHOD

A round-bottom or an Erlenmeyer flask with an Anschütz adapter and a reflux condenser, the end of which is closed off with a CaCl<sub>2</sub> tube, is used. The flask is filled with 100 g. of PCl<sub>3</sub> and 24 g. of powdered S and heated to boiling on a steam bath. As soon as the mixture is boiling vigorously, 3 to 5 g. of finely powdered anhydrous AlCl<sub>3</sub> is added. The sulfur dissolves quickly with vigorous to violent boiling. The flask must sometimes be somewhat cooled. Toward the end of the reaction, which occurs within 5 to 10 minutes, the liquid becomes orange-yellow. At that point the boiling ceases, indicating that all of the  $PCl_3$  is transformed into  $PSCl_3$ .

The cooled liquid is now poured into a large separatory funnel, a large amount of water is added, and the funnel is carefully shaken to avoid too heavy an emulsion. This dissolves the  $AlCl_3$ ,  $PCl_5$ ,  $H_3PO_3$  and HCl, producing an immediate decolorization of the product. The  $PSCl_3$  settles out as the bottom. It is separated, dried with  $CaCl_2$  and distilled.

The yield is as high as  $120 \text{ g}_{\bullet} (97\%)$ .

SYNONYM:

Phosphorus sulfochloride.

PROPERTIES:

Colorless, mobile liquid; fumes in air; sharp odor, not disagreeable when diluted; lachrymator. With water, decomposes slowly in the cold, quickly when heated, to give HCl,  $H_2S$  and  $H_3PO_4$ . On heating with sodium hydroxide,  $Na_3PO_3S$  is formed (see p. 569). Miscible with  $CS_2$ .

M.p. -35°C, b.p. 125°C (corr.); d 1.668.

**REFERENCES:** 

- I. German Patent 675,303, Class 12 i, Group 31, May 5, 1929 (Inventor: G. Schrader).
- II. F. Knotz. Österr. Chemiker-Z. <u>50</u>, 128 (1949).

### Phosphoryl (V) Bromide

#### POBr<sub>3</sub>

According to Hönigschmid and Hirschbold-Wittner, the reaction of  $PBr_5$  with  $P_2O_5$ , proposed by Berger, is the best procedure for the preparation of  $POBr_3$ :

A round-bottom flask, joined to a reflux condenser with a ground joint, is the reactor. A mixture of  $PBr_5$  and  $P_2O_5$  (mole ratio 5 : 1), with a small excess of the latter [e.g., 250 g. of  $PBr_5$  and

20 g. of  $P_2O_5$ —preferably from a new package] is heated in an oil bath, with the temperature gradually increased to 150°C. Care must be taken to prevent escape of the bromine. The reaction is complete after five hours. Then 10 g. of  $Br_2$  and a corresponding quantity of  $P_2O_5$  are added to the molten product. The mixture is refluxed for seven hours at 150°C. This oxidizes the intermediate PBr<sub>3</sub> to PBr<sub>5</sub> and transforms the latter into POBr<sub>3</sub>. The final product is distilled at 12 mm. A tube containing NaOH must be inserted between the aspirator and the distillation apparatus. The first cut contains  $Br_2$  and some PBr<sub>3</sub>. The completely colorless POBr<sub>3</sub> is obtained almost quantitatively. It is best to cool the receiver with an ice-salt mixture. The yield is 200 g. (73%, based on the PBr<sub>5</sub> used).

The traces of  $PBr_3$  can only be removed by fractionating the  $POBr_3$  six times in high vacuum.

PROPERTIES:

Very sensitive to elevated temperature, at which it decomposes with yellowing. For this reason it should never be melted with a flame, but only with hot water.

Large, flaky crystals. M.p. 55°C, b.p. 193°C; d 2.82.

Decomposes slowly in water, forming  $H_3PO_4$  and HBr. Soluble in ether.

**REFERENCES:** 

O. Hönigschmid and F. Hirschbold-Wittner. Z. anorg. allg. Chem. 243, 355 (1940).

E. Berger. Compt. Rend. Hebd. Séances Acad. Sci. 146, 400 (1908).

### Thiophosphoryl (V) Bromide

#### PSBr<sub>3</sub>

Thiophosphoryl bromide can be obtained by the reaction of phosphorus pentasulfide with phosphorus pentabromide:

$$P_2S_5 + 3 PBr_5 = 5 PSBr_3$$
  
 $\frac{1}{10}: 22.2 \quad 129.2 \quad 151.4$ 

The reaction vessel is a distillation flask provided with a  $P_2O_5$  drying tube. The flask is charged with 31 g. of dry, red P and cooled in a bath. Then 400 g. of Br is added, followed by 100 g. of  $P_2S_5$ . The mixture is then heated for two hours on a water bath and finally with an open flame, until completely

liquid. The  $PSBr_3$  is distilled at 25 mm. and the fraction distilling between 120 and  $130^{\circ}C$  is collected. The yield, based on P, is 80-85%.

To purify the crude product, it is added to twice its volume of distilled water plus a few drops of a 10% solution of KBr. The mixture is heated on a water bath until liquid. A slow stream of compressed air is bubbled through the mixture for a few minutes, the water is decanted, and any sulfur which might have settled out is removed. The yellow layer of PSBr<sub>3</sub> is allowed to crystallize, the water traces are removed, and the compound is dried over  $P_2O_5$ . The yield of pure PSBr<sub>3</sub> is 60%.

PROPERTIES:

Dissolved in PBr<sub>3</sub>, it crystallizes in yellow, regular octahedra. The melt hardens to fibers.

M.p.  $38^{\circ}$ C, b.p.  $212-215^{\circ}$ C (dec.),  $125-130^{\circ}$ C (25 mm.); d ( $17^{\circ}$ C) 2.85.

Readily soluble in ether,  $CS_2$ ,  $PCl_3$  and  $PBr_3$ . Fairly stable in the presence of water, and even forms a monohydrate with it [m.p. 35°C, d (18°C) 2.794]. The hydrate decomposes at the melting point into its constituents; the same happens on dissolving in  $CS_2$ . The water may be removed with  $CaCl_2$ .

REFERENCE:

H. S. Booth and C. A. Seabright in: W. C. Fernelius, Inorganic Syntheses, Vol. 2, p. 153, New York-London, 1946.

### Diphosphoric Acid Tetrachloride

### $P_2O_3Cl_4$

According to Geuther and Michaelis,  $P_2O_3Cl_4$ , together with other phosphorus oxychlorides, can be obtained by the reaction of  $PCl_3$  with  $N_2O_4$ . Klement et al. have improved this procedure and have confirmed, among other things, the simultaneous formation of tetraphosphoryl decachloride  $P_4O_4Cl_{10}$ . Because of the complexity of the reaction, no stoichiometric equation can be written (I). The procedure of Huntly, based on heating a mixture of  $POCl_3$  and and  $P_4O_{10}$  was improved by Grunze (II).

I. Nitrogen oxides  $(N_2O_3, \text{ or } NO + NO_2)$  are produced in flask  $\alpha$  (see Fig. 196) over a period of about four hours by dropping about 550 ml. of 68%  $H_2SO_4$  onto 500 g. of ice-cooled crystalline NaNO<sub>2</sub> (Hofmann and Zedtwitz procedure). The jacket of reflux condenser b is filled with lukewarm water. Sufficient dry  $O_2$  is



Fig. 193. Preparation of  $P_2O_3Cl_4$  and  $P_4O_4Cl_{10}$ : a) prereaction flask; b) reflux condenser; c) empty trap; d) trap filled with glass wool; e) mixing trap; f) measuring trap; g) reaction flask;  $h_1 - h_2$ ) stopcocks.

mixed with the nitrogen oxides to assure that all N<sub>2</sub>O<sub>3</sub> is converted to  $N_2O_4$ . The  $O_2$  is mixed at stopcock  $h_2$ . The gaseous mixture is condensed in mixing trap f, cooled with liquid nitrogen. With stopcocks  $h_1$  and  $h_4$  closed and stopcocks  $h_2$ ,  $h_3$ ,  $h_5$  and  $h_6$  open the  $N_2O_4$  (about 200 g.) is distilled from f by heating the latter to 22°C. It is collected in 1000 g. of freshly distilled PCl<sub>3</sub> in flask q. The addition proceeds with vigorous stirring. At the same time, a very slow stream of O<sub>2</sub> is passed through. For greater safety, an empty trap is inserted between flask q and stopcock  $h_5$ . During the reaction (10-11 hours), flask q must be well cooled to -25 to  $-21^{\circ}$ C (by means of a trichloroethylene-Dry Ice bath), because this influences the yield. The dark red liquid is allowed to stand overnight in flask q, with stopcock  $h_5$  closed and stopcock  $h_6$  open. By heating flask q in a water bath to no more than 30°C, most of the gaseous product (particularly NOC1) is driven off. The mixture must be stirred during the distillation. The yellow-red liquid is then placed in a distilling flask and distilled at 11 mm. Nitrogen oxides and NOC1 distill first, then POC1<sub>3</sub> (about 850 g.). The receiver must be cooled with ice-salt mixture. It is best to insert several liquid-nitrogen-cooled traps between the receiver and the pump. The light-brown residue is distilled at 10<sup>-3</sup> mm., with the flask heated on an oil bath. The bath temperature is slowly raised to 120°C. Residual POCl<sub>a</sub> distills first; then a colorless liquid follows from 35°C on. A dark mass (a few grams) remains in the flask.

The distillate is fractionated very slowly at  $10^{-3}$  mm. and the fractions boiling between 35 and 50°C (I), between 50 and 60°C (II)

and between 60 and 70°C (III) are collected separately. Fraction II is divided into lower boiling ( $\alpha$ ) and higher boiling portions (b). Portion  $\alpha$  is added to I and portion b to III. Fraction I is then refractionated, and almost completely pure  $P_2O_3Cl_4$  (90-100 g.) is obtained between 36 and 38°C. By fractionation of III, almost pure  $P_4O_4Cl_{10}$  is obtained between 63 and 68°C (about 70 g.). To obtain very pure material, the crude must be refractionated. The high-vacuum boiling temperatures given below depend on the dimensions of the apparatus and the rate of the condensation. At 10-12 mm.,  $P_2O_3Cl_4$  boils at 90-92°C and  $P_4O_4Cl_{10}$  at 137-138°CPure  $P_4O_4Cl_{10}$  solidifies even in the condenser (colorless crystals, m.p. 38°C). It is therefore best to fill the condenser with water at 45°C.

II. A mixture of  $POCl_3$  and  $P_4O_{10}$  (mole ratio 8:1) is heated in a combustion tube for 48 hours at  $200^{\circ}C$ . The  $POCl_3$  is distilled from the product at 12 mm. (the flask is placed in hot water). The  $P_2O_3Cl_4$  is then quantitatively distilled off at 12 mm., with the flask placed in a sand bath at  $250^{\circ}C$ . A repeat distillation at even lower pressure, using a boiling water bath, yields completely pure  $P_2O_3Cl_4$ . The yield is about 30%.

SYNONYMS:

Pyrophosphoryl chloride, dichlorylphosphoric acid anhydride.

PROPERTIES:

Formula weight 251.8. Colorless liquid. Fumes only after standing for some time in air. However, reacts vigorously with water. Dichlorylphosphoric acid HPO<sub>2</sub>Cl<sub>2</sub> can be obtained by careful reaction with water below  $-30^{\circ}$ C (Grunze). Soluble in PCl<sub>3</sub>, POCl<sub>3</sub>, SOCl<sub>2</sub>, C<sub>6</sub>H<sub>6</sub> and other hydrocarbons, ether and nitrobenzene.

M.p.  $-16.5^{\circ}$ C, b.p. (in the vacuum of a mercury diffusion pump)  $47^{\circ}$ C.  $d_{15}^{15}$  1.82.

**REFERENCES:** 

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- II. O. N. Huntly. J. Chem. Soc. (London) 59, 202 (1891); H. Grunze.
   Z. anorg. allg. Chem. 296, 63 (1958; 298, 152 (1959).

### Diphosphorus Tetraiodide

#### $P_2I_4$

Diphosphorus tetraiodide is formed exothermically from the elements mixed in the stoichiometric ratio. The reaction can be moderated with CS<sub>2</sub>, which also serves as a solvent:

> $2 P + 2 I_2 = P_2 I_4$ 61.95 507.64 569.59

According to the procedure of Germann and Traxler, improved by Baudler, a solution of 6.2 g. of white P in 100 ml. of CS<sub>2</sub> (carefully purified and distilled over  $P_2O_5$ ) is poured through a fritted glass filter into a ground joint flask. This treatment removes small amounts of suspended contaminants. A filtered solution of 50.77 g. of sublimed iodine in 500 ml. of pure CS<sub>2</sub> is added in portions with shaking, making certain that the flask is opened for a short period only. Before adding a fresh portion, wait until the initially dark red-brown mixture becomes transparent and bright red. The 50-ml. CS<sub>2</sub> rinse of the I<sub>2</sub> container is also added. The mixture is then allowed to stand in the dark in a well stoppered flask for twelve hours to complete the reaction. The clear orange-red solution is then poured into a suction flask which is connected through a CaCl<sub>2</sub> tube to an aspirator. It is then concentrated. Care should be taken that this operation does not take too long and that the solvent is kept boiling gently all the time. The rate of boiling is controlled through the steam bath temperature. After a short time, crystals of  $P_2I_4$  are deposited on the wall and are washed down by swirling the liquid. When the solution is concentrated to 60-80 ml., the crystals are quickly suction-filtered on a fritted glass filter and gently crushed with a glass rod. The filter with contents is immediately placed in a vacuum desiccator and evacuated to 20 mm, while protecting the product from moisture with a CaCl<sub>2</sub> drying tube. The crystals are then dry enough to be easily pulverized without smearing. Longer drying is harmful as it favors slight decomposition of the  $P_2I_4$ . Since considerable decomposition takes place even after a short time in a desiccator over CaCl<sub>2</sub>, the product is stored in sealed ampoules or in carefully closed bottles with ground glass stoppers. The yield is 37-41 g., or 65-75% of theoretical. With careful operation, the mother liquor can be reused for a new batch.

SYNONYM:

Phosphorus diiodide.

PROPERTIES:

Thin, red prisms. M.p.  $125.5^{\circ}$ C Decomposes with water to  $H_3PO_3$ , PH and HI.

REFERENCES:

- A. Michaelis and M. Pitsch. Liebigs Ann. Chem. 310, 66 (1900).
- F. E. E. Germann and R. N. Traxler. J. Amer. Chem. Soc. <u>49</u>, 307 (1929).
- M. Baudler. Z. Naturforschg. 13b, 266 (1958).

### Phosphorus (III) Iodide

#### PI<sub>3</sub>

Phosphorus (III) iodide can be prepared either from red or white P, dissolved in  $CS_2$ , by reaction with a solution of  $I_2$  in  $CS_2$ :

 $\begin{array}{c} P + 1^{1} /_{2} I_{2} = PI_{3} \\ 31 & 380.7 & 411.7 \end{array}$ 

As Germann and Traxler have established, very carefully purified  $CS_2$  must be used. Impure  $CS_2$ , containing S, causes the formation of sulfurated  $PI_3$ , the presence of which lowers the melting point.

Purification of the reagents: a) Iodine is purified by grinding with KI and subliming. b) Red phosphorus should have been exposed to the air as little as possible. It is washed with freshly purified  $CS_2$  in order to remove white P. Commercial white phosphorus is usually pure enough. c) Very pure carbon disulfide is shaken in a glass-stoppered flask with portions of pure Hg until the free S is removed. This requires long contact with several portions of Hg. The treatment should be continued until only a slight discoloration is evident on the bright surface of freshly added mercury after several minutes of contact with the  $CS_2$ . The latter is then filtered, distilled and used immediately.

#### I. PREPARATION WITH RED PHOSPHORUS

The required amount of  $I_2$  is dissolved in  $CS_2$ , and excess red P is added. After the disappearance of free  $I_2$ , the dark-red, opaque solution is filtered from unreacted P and the  $CS_2$  is distilled off on a sand bath until crystals appear. The solution is then allowed to cool, the supernatant liquid is decanted, and the remaining crystals are gently warmed.

#### **II. PREPARATION WITH WHITE PHOSPHORUS**

Two solutions in  $CS_2$  are prepared. One contains 1 g. of white P and the other 12.27 g. of  $I_2$ . The solutions are mixed without loss, and the mixture is processed further as under I.

PROPERTIES:

Dark red, columnar crystals. M.p. 61°C. Quick decomposition in moist air and must therefore be stored over CaCl<sub>2</sub>.

REFERENCE:

F. E. E. Germann and R. N. Traxler. J. Amer. Chem. Soc. <u>49</u>, 307 (1927).

### Phosphorus (V) Oxide

### $P_2O_5 (P_4O_{10})$

Purification of the commercial product. Ordinary commercial  $P_2O_5$  usually contains lower oxides of P, especially  $P_2O_3$ , and sometimes also white P. When  $P_2O_5$  is used as a drying agent, these impurities sometimes exert a harmful influence because of their reducing action. To test for the lower oxides, P2O5 is dissolved in water, forming a solution which easily reduces a 10% AgNO<sub>3</sub> solution, and a Hg(II) salt solution on boiling. If lower P oxides are present a distinct odor of PH<sub>a</sub> is given off when the aqueous solution is evaporated and then moderately warmed. To prepare pure P<sub>2</sub>O<sub>5</sub>, the commercial product is sublimed in a stream of well-dried oxygen at bright red heat, according to Finch and Peto, and also Whitaker. A T-shaped iron tube is used for this purpose (see Fig. 194); it is connected to a glass tube. The apparatus, particularly the iron tube, must be thoroughly cleaned and dried. The commercial product to be sublimed is gradually added from a and trapped in collecting bulb b. From 200 g. of impure  $P_2O_5$ , about 70 g. of pure  $P_2O_5$  can be obtained in two hours (cf. also Part I, p. 81).

Modifications of phosphorous pentoxide. Phosphorus pentoxide forms three solid modifications, of which the metastable M form is the ordinary commercial  $P_2O_5$ . This modification crystallizes as rhombohedra with a molecular lattice ( $P_4O_{10}$ ) and sublimes readily at 250°C and 10 mm. (Glixelli and Boratynski). Above 260°C and even more quickly above 500°C, form M changes into form R. The latter crystallizes in a three-dimensional atomic lattice of  $PO_4$  tetrahedra and is less volatile. A form S, which



Fig. 194. Purification of  $P_2O_5$  by sublimation. a) Starting material; b) collecting bulb for purified material; c) electric furnace.

crystallizes in a layer lattice, and a few vitreous forms are also known (De Decker and McGillavry; Hill, Faust and Hendricks).

To prepare the stable modification, crystals of the M modification, obtained by sublimation in a stream of dry  $O_2$  at about 320°C, are placed under a stream of dry  $O_2$  in a Vycor tube 2 × 23 cm. The latter is fused shut after evacuation. After heating for five days at 500-530°C in a horizontal position, the section of the tube containing the best crystals is heated for 2.5 hours at 350-400°C, while the other end remains at room temperature.

SYNONYM:

Tetraphosphorus decaoxide.

**PROPERTIES:** 

Modification M is brittle; R and S form hard crystals, which deliquesce after a few hours in the air and undergo considerable swelling in water. The density of R is 2.72, that of M 2.30 (calculated from x-ray data).

**REFERENCES:** 

- G. I. Finch and R. H. K. Peto. J. Chem. Soc. (London) <u>121</u>, 692 (1922),
- H. Whitaker. J. Chem. Soc. (London) 127, 2219 (1925).
- S. Glixelli and K. Boratynski. Z. anorg. allg. Chem. 235, 225 (1938).
- H. C. J. de Decker and C. H. McGillavry. Rec. Trav. Chim. Pays-Bas 60, 153 and 413 (1941); Nature 164, 448 (1949).
- W. L. Hill, G. T. Faust and S. B. Hendricks. J. Amer. Chem. Soc. 65, 794 (1943).

# Orthophosphoric Acid H<sub>3</sub>PO<sub>4</sub>

#### CRYSTALLINE ORTHOPHOSPHORIC ACID

According to Simon and Schulze, very pure, crystalline  $H_3PO_4$  may be prepared by evaporating 83% phosphoric acid in high vacuum to crystallization.



Fig. 195. Evaporation of phosphoric acid in vacuum. A) Pt-Au alloy dish; b) condensation trap.

A platinum-gold alloy dish is fastened by melting in a glass flask (see Fig. 195). The dish is filled with 83% phosphoric acid, which is then concentrated to 99% at 80°C and 1 mm. The procedure The dehydration is continuously checked by takes one week. weighing the liquid-nitrogen-cooled trap. The highly concentrated acid obtained is introduced into fritted glass container A of the crystallization apparatus (see Fig. 196), which is then evacuated. By sharp cooling of the lower section of A with Dry Ice-alcohol mixture, a seed crystal is produced. The apparatus is then immediately transferred to a thermostat held at 38°C. The temperature is then gradually lowered until the seed crystal continues to grow slowly. Usually the temperature is not allowed to drop below 35°C. In two to three days, a large part of the acid solidifies to a loose, crystalline network. Then dried air is introduced through the fritted glass vessel  $F_1$ , which is filled with pumice chips and  $P_2O_5$ . The liquid part of the acid which collects in B is removed by suction through C. The separation of the crystals from

the liquid is greatly facilitated by the behavior of  $H_3PO_4$  crystals, which are not wetted by the liquid acid. After a second evacuation the crystals are melted and the crystallization is repeated at a temperature about 0.5° higher.

SYNONYM:

Phosphoric acid.

PROPERTIES

The crystals remaining after the second suction drying yield no precipitate of  $Zn_2P_2O_7$  when reacted with  $ZnSO_4$  in an acetic acid solution. They contain 99.6% to 100.1% H<sub>3</sub>PO<sub>4</sub>. M.p. 41.5°C; d(18°C) 1.834.



Fig. 196. Preparation of crystalline orthophosphoric acid. A) Fritted glass container; B) collecting vessel for the mother liquor; C) suction connection;  $F_1, F_2$ ) fritted glass disks.

REFERENCE:

A. Simon and G. Schulze. Z. anorg. allg. Chem. 242, 322 (1939).

### Sodium Dihydrogen Phosphate

### $NaH_2PO_4 \cdot 2H_2O$

According to Beans and Kiehl,  $NaH_2PO_4$  can be obtained as the dihydrate by crystallization from cold aqueous alcoholic solution.

Very pure  $NaH_2PO_4$  is recrystallized three times from a mixture of equal volumes of distilled water and 95% alcohol. The crystallization proceeds in an ice bath. The crystals are suctiondried, washed three times with absolute alcohol and three times with absolute ether, and allowed to stand for a short time in the air to evaporate the ether. The salt must be stored in fused vessels or water is lost because of the high dissociation pressure.

PROPERTIES:

Formula weight 156.02. Rhombic-disphenoidal crystals. d 1.915. Begins to melt at  $60^{\circ}$ C.

REFERENCE:

H. T. Beans and S. J. Kiehl. J. Amer. Chem. Soc. 49, 1878 (1927).

544

### Potassium Phosphate

### K<sub>3</sub>PO<sub>4</sub> · 8 H<sub>2</sub>O

Pure  $K_3PO_4$  cannot be obtained by recrystallization. According to Simon and Schulze, the octahydrate can be prepared in a sufficiently pure form, using a procedure reported by Jänecke. A solution of 300 g. of C. P.  $K_3PO_4$  in 180 ml. of water is prepared. After bubbling ammonia through this solution, for 2-3 hours, potassium phosphate octahydrate precipitates in large amounts. The salt is suction-dried in well-dried air.

PROPERTIES:

Formula weight 356.4. Flat, rectangular flakes. M.p.  $45.1^{\circ}$ C. Solubility in water (0°C): 43.7; (25°C) 50.8; (45.1°C) 59.7% K<sub>3</sub>PO<sub>4</sub>.

**REFERENCES:** 

E. Jänecke. Z. phys. Chem. <u>127</u>, 75 (1927);
A. Simon and G. Schulze. Z. anorg. allg. Chem. <u>242</u>, 331 (1939).

### **Hydroxyapatite**

Ca10(PO4)6(OH)2

Crystalline hydroxyapatite is formed on very slow precipitation from extremely dilute solutions:

 $\begin{array}{r} 10 \operatorname{Ca}(\mathrm{NO}_3)_2 \,+\, 6 \,\mathrm{KH}_2 \mathrm{PO}_4 \,+\, 14 \,\mathrm{NaOH} \,=\, \mathrm{Ca}_{10}(\mathrm{PO}_4)_6(\mathrm{OH})_2 \,+\, 6 \,\mathrm{KNO}_3 \\ {}^{1}\!/_{10} \colon 164.1 \quad 81.7 \quad 56.0 \quad 100.5 \quad 60.7 \\ & +\, 14 \,\mathrm{NaNO}_3 \,+\, 12 \,\mathrm{H}_2 \mathrm{O} \\ 119.0 \quad 21.6 \end{array}$ 

Hayek, Müllner and Koller (II) obtained well-formed single crystals of hydroxyapatite as needles up to 0.03 mm. long by digesting calcium phosphate (or  $CaHPO_4 \cdot 2H_2O$ ) with  $CO_2$ -free, very pure water. Single crystals up to 2 mm. long were obtained hydrothermally in sodium hydroxide (Hayek, Böhler, Lechleitner and Petter).

I. Two liters of water are kept for a week on a hot plate at about  $95^{\circ}$ C, with continuous replacement of the evaporated water. Every day five drops each of the following solutions are added: a) 16.9 g. of Ca(NO<sub>3</sub>)<sub>2</sub> in one liter of CO<sub>2</sub>-free water, b) 5.84 g. of KH<sub>2</sub>PO<sub>4</sub> (Sörensen method for enzyme studies) in one liter of CO<sub>2</sub>-free

water. With the help of a few drops of  $CO_2$ -free 0.1N NaOH, the solution is kept neutral to bromthymol blue, and some seed crystals of hydroxyapatite, precipitated from a concentrated solution, are added. After four days, crystals begin to separate. These are just visible to the naked eye and grow somewhat larger during the next three days. They are filtered and washed with water.

II. The precipitation product from mixing stoichiometric quantities of solutions of  $Na_3PO_4$  and  $Ca(NO_2)_2$  or that from mixtures of  $Ca(OH)_2$  and  $H_3PO_4$  (or  $CaHPO_4 \cdot 2H_2O$ ) is boiled about 20 times, each time for an hour, with  $CO_2$ -free, pure water. The water is renewed each time. The weight ratio of the sediment to the water should not exceed 1:30. If  $CaHPO_4 \cdot 2H_2O$  is used, it is recommended that a larger amount of water (about 100:1) be used for the first heating.

### PROPERTIES:

Hexagonal needles and druses. Only slightly soluble in water.

### **REFERENCES:**

- I. W. Rathje. Ber. dtsch. chem. Ges. 74, 347 (1941).
- E. Hayek, F.Müllner and K. Koller. Monatsh. Chem. <u>82</u>, 959 (1951);
   E. Hayek, W. Böhler, J. Lechleitner and H. Petter. Z. anorg. allg. Chem. <u>295</u>, 241 (1958).

## **Condensed Orthophosphates**

### DIPHOSPHORIC ACID, H4P2O7

Crystalline diphosphoric acid. The reaction discovered by Geuther:

$$5 H_{3}PO_{4} + POCl_{3} = 3 H_{4}P_{2}O_{7} + 3 HCl$$

$$490.0 153.3 533.9 109.4$$

is recommended by Partington and Wallsom for obtaining very pure  $H_4P_2O_7$ .

A mixture of  $H_3PO_4$  and  $POCl_3$  is carefully evaporated in a platinum dish at 180°C. The residue is allowed to crystallize in a cooled desiccator.

SYNONYM:

Pyrophosphoric acid.

PROPERTIES:

Vitreous crystals. M.p. 61°C. Soluble without change in ice water; gradually forms orthophosphoric acid at higher temperature.

**REFERENCES:** 

- A. Geuther. J. prakt. Chem. [2] 8, 359 (1874).
- J. R. Partington and H. E. Wallsom. Chem. News <u>136</u>, 97 (1928), as reported by Chem. Zentr. 1928, I, 1936.

PENTASODIUM TRIPHOSPHATE, Na<sub>5</sub>P<sub>3</sub>O<sub>10</sub>

Pentasodium triphosphate is prepared by annealing a quenched melt of solid  $(NaPO_3)_n$  (Graham's salt) and  $Na_4P_2O_7$  between 300 and 500°C (Huber):

 $\begin{array}{r} N_{a}PO_{3} + N_{a_{4}}P_{2}O_{7} = N_{a_{5}}P_{3}O_{10} \\ 102.0 & 266.0 & 367.9 \end{array}$ 

The hexahydrate crystallizes from an aqueous solution of the reaction products (method I).

According to Thilo and Ratz (method II), the salt can be quantitatively prepared in solution by cleavage of sodium trimetaphosphate (see p. 552) with the equivalent amount of alkali according to the equation

 $\begin{array}{rl} Na_{3}P_{3}O_{9} \,+\, 2\,NaOH \,=\, Na_{5}P_{3}O_{10} \,+\, H_{2}O \\ _{305.9} & _{80.0} & _{367.9} & _{18.0} \end{array}$ 

I. An intimate mixture of 102 g of  $(\text{NaPO}_3)_n$  and 266 g of  $\text{Na}_4\text{P}_2\text{O}_7$ is melted. The melt is quenched, pulverized, and pressed into tablets of 2-3 g., and these are annealed for eight hours at 500 to  $525^{\circ}\text{C}$ . Then 10 g of the annealed reaction product is dissolved in water. The solution is evaporated over  $\text{H}_2\text{SO}_4$  and an unstable octahydrate crystallizes out. It is dried over  $\text{P}_2\text{O}_5$  and thus transformed to the hexahydrate. The latter is stable. The hexahydrate can also be produced by precipitation from the solution with alcohol.

II. Sodium trimetaphosphate (10 g.) is placed in a porcelain dish and covered with a solution of 2.611 g. of NaOH in 60 ml. of water. This is heated on a steam bath until dissolution occurs. After 2-3 minutes the reaction is complete as indicated by the fact that the solution yields a pure white precipitate with AgNO<sub>3</sub>. Prolonged heating should be avoided, as it causes hydrolysis of the triphosphate to the orthophosphate. The solid salt is obtained from the solution by precipitating with alcohol or evaporating in vacuum over  $H_2SO_4$  at about 40°C. The yield is 100%.

The salt is pure if its solution gives a white precipitate with  $AgNO_3$  which is readily soluble in dilute sulfuric acid and in dilute ammonia. If the precipitate is brownish, the reaction was not complete; a yellowish precipitate indicates the presence of products of hydrolysis.

SYNONYM:

Sodium tripolyphosphate.

PROPERTIES:

Na  ${}_{5}P_{3}O_{10} \cdot 6H_{2}O$  formula weight 476.02. Rectangular parallelepipeds. d 2.12. On dehydration at 100°C, only 5/6 of the water of crystallization is given up quickly, while the last molecule of water acts destructively according to the equation: Na  ${}_{3}P_{3}O_{10} + H_{2}O =$ Na  ${}_{4}P_{2}O_{7} + NaH_{2}PO_{4}$ . Above 120°C the salts react to give up water and form pentasodium triphosphate: Na  ${}_{4}P_{2}O_{7} + NaH_{2}PO_{4} =$ Na  ${}_{5}P_{3}O_{10} + H_{2}O$  (according to Thilo). Anhydrous sodium triphosphate forms two monoclinic crystalline forms with differing densities: form I, d 2.52, form II, d 2.59 (Dymon and King). The anhydrous compound melts incongruently at 622°C, forming Na  ${}_{4}P_{2}O_{7}$ crystals and a melt containing 49.5 weight % Na  ${}_{4}P_{2}O_{7}$ .

REFERENCES:

- H. Huber. Angew. Chem. 50, 323 (1937).
- P. Bonneman. Compt. Rend. Hebd. Séances Acad. Sci. 204, 433 (1937).
- P. Bonneman and M. Bassière. Compt. Rend. Hebd. Séances Acad. Sci. <u>206</u>, 1379 (1938).
- E. P. Partridge, V. Hicks and G. W. Smith. J. Amer. Chem. Soc. 63, 454 (1941) as reported by Chem. Zentr. <u>1941</u> II, 548.
- G. W. Morey and E. Ingerson. Amer. J. Sci. 242, I (1944) as reported by Chem. Zentr. 1945 I, 1225.
- E. Thilo and R. Rätz. Z. anorg. allg. Chem. 258, 33 (1949).
- E. Thilo. Sitzungsber. Deutsche Akad. Wiss. Berlin, Kl. Mathemat. u. allg. Naturwiss. 1952, No. 1.
- J. J. Dymon and A. J. King. Acta Cryst. (London) 4, 378 (1951).

HEXASODIUM TETRAPHOSPHATE, Na<sub>6</sub>P<sub>4</sub>O<sub>13</sub>

Hexasodium tetraphosphate is formed by careful hydrolysis of sodium tetrametaphosphate (see p. 553) (method of Thilo and Ratz):

$$\begin{array}{rrrr} Na_4P_4O_{12} + 2 NaOH = Na_6P_4O_{13} + H_2O \\ (4 H_2O) & (aq) \\ 479.9 & 80.0 & 18.0 \end{array}$$

A solution of 4.8 g. of  $Na_4P_4O_{12} \cdot 4H_2O$  in 85 ml. of hot water is prepared, and after cooling, a solution of 0.8 g. of NaOH in some water is added. The mixture is kept over  $H_2SO_4$  in a vacuum desiccator, placed in an oven at 40°C. The rate of evaporation should be such that after about 100 hours no less than 8 to 10 ml. remains. The solution is then diluted to 30 ml. and acetone is added. The colorless oil formed is dissolved in some water and reprecipitated with acetone.

**PROPERTIES:** 

 $Na_{g}P_{4}O_{13} \cdot H_{2}O$ . Formula weight of the anhydrous salt is 469.85. The crystalline salt cannot be obtained. On standing for four weeks, the dried oil is partially transformed to  $Na_{3}HP_{2}O_{7}$ .  $H_{2}O$ . Heating of this mixture to  $200^{\circ}C$  for two hours produces a quantitative yield of  $Na_{3}HP_{2}O_{7}$ .

**REFERENCE:** 

E. Thilo and R. Rätz. Z. anorg. allg. Chem. 260, 255 (1949).

### **Polyphosphates**

MADRELL'S SALT, SODIUM POLYPHOSPHATE (NaPO<sub>3</sub>)x

According to V. Knorre, sodium polyphosphate, called Madrell's salt, is obtained by heating the residue from evaporation of a solution of NaNO<sub>3</sub> and phosphoric acid.

A solution of 20 g. of NaNO<sub>3</sub> in 25 ml. of water is prepared, mixed with 42 ml. of phosphoric acid (d 1.3) and evaporated on a water bath. The residue is then heated for four hours at  $330^{\circ}$ C and the melt is extracted with water. The salt is obtained as a practically insoluble white powder. The yield is about 95%. For unknown reasons, the preparation sometimes proves unsuccessful.

PROPERTIES:

Difficultly soluble in acetic acid. Readily soluble in cold dilute sulfuric acid, cold dilute nitric acid, and hot dilute hydrochloric acid to form orthophosphate. According to Thilo and Plaetschke, Madrell's salt is a chainlike, polymerized polymetaphosphate. According to Partridge, Hicks and Smith, it exists in two modifications, which, according to Thilo, differ in the lengths of their chains.

**REFERENCES:** 

- R. Madrell. Liebigs Ann. Chem. 61, 63 (1847).
- G. von Knorre. Z. anorg. allg. Chem. 24, 397 (1900).
- E. Thilo and I. Plaetschke. Z. anorg allg. Chem. 260, 297 (1949).
- E. P. Partridge, V. Hicks and G. W. Smith. J. Amer. Chem. Soc. 63, 454 (1941).
- E. Thilo. Angew. Chem. 63, 508 (1951).

### GRAHAM'S SALT (NaPO<sub>3</sub>)y

The ordinary commercial material is obtained as a transparent glass in the Graham method by heating NaH<sub>2</sub>PO<sub>4</sub> and quickly quenching the melt. It is still often incorrectly called "sodium hexametaphosphate" but, according to Karbe and Jandev, it is in no way a simple, well-defined hexaphosphate. Rather, most of the preparations are highly polymerized. The degree of polymerization depends on the heating temperature and reaches a maximum at about  $1100^{\circ}$ C. Thereafter, it begins to fall off, as can be seen by measuring the anion weight in the determination of the dialysis coefficient. The maximum corresponds to an anion weight of 3460, which (with certain assumptions) corresponds to about 44 PO<sub>3</sub> groups in the anion. Karbe and Jander give the following directions for a uniform and reproducible preparation:

A definite (always the same) amount of  $NaH_2P_4 \cdot 2H_2O$  is placed in a platinum dish and, after a two-hour dehydration at somewhat above 200°C, is placed in an electric furnace set at the desired temperature. The run itself starts from time at which the temperature, after an initial drop, regains constancy. This occurs approximately one half hour after placing the dish in the furnace. At the end of a four-hour heating cycle, the melt is removed from the furnace and quenched as quickly as possible by pouring into a large iron dish filled with dry CCl<sub>4</sub>. The dish is externally cooled with ice-salt mixture. To speed up the cooling process, several pieces of Dry Ice are placed in the  $CCl_{4}$ , before the addition of the melt. They are soon coated with a layer of solid CCl<sub>4</sub>. The rate of cooling is increased considerably by the melting of this layer which follows on addition of the hot mass and by further evaporation of the CO<sub>2</sub>. The vitreous product is filtered off. It usually breaks up spontaneously because of the great internal stresses and the crushing can be completed by slight agitation. The adsorbed CCl<sub>4</sub> is removed in vacuum. The pieces of glassy product are immediately sealed in an air-tight container.

#### PROPERTIES:

Extremely hygroscopic; becomes moist and sticky even after standing in the air for a short time. Dissolves slowly in cold water but is quite soluble in warm water between 30 and  $50^{\circ}$ C. According to Bronnikov, solubility ( $20^{\circ}$ C) 973.2 g./liter; ( $80^{\circ}$ C) 1744 g./liter. Its solutions give precipitates with Mg, Ca, Ba, Pb and Ag salts. These precipitates are soluble in an excess of the polyphosphate. Concentrated NaCl solution and alcohol flocculate the solutions, gradually forming a viscous, oily mass. No definite melting point; with careful heating, the material starts to liquefy somewhat above  $600^{\circ}$ C. **REFERENCES:** 

- T. Graham. Pogg. Ann. <u>32</u>, 64 (1834).
- K. Karbe and G. Jander. Kolloid-Beihefte 54, 80-91 (1942).
- A. Kh. Bronnikov. Zh. Prikladnoy Khimii <u>12</u>, 1287 (1939) (cited in Karbe and Jander).

### KURROL'S SODIUM POLYPHOSPHATE, (NaPO<sub>3</sub>)<sup>z</sup>

The preparation starts with the production of seed crystals. Thus 85 g. of  $Na_2HPO_4$  and 15 g. of  $NH_4H_2PO_4$  are heated at 800-900°C until all water and  $NH_3$  are removed. Then the mixture is allowed to cool and kept at a constant temperature between 650 and 550°C for a few hours. The melt solidifies almost completely to a fibrous product, which is still somewhat impure, since an excess of phosphoric acid was used. The fibrous mass is pulverized, washed several times with water and dried with alcohol and ether.

Then, in the Pascal method, a melt of Graham's salt is allowed to cool to  $600^{\circ}$ C, the seed crystals are strewn on its surface, and the melt is kept for another half hour at 550°C, during which it solidifies to a completely pure material.

#### PROPERTIES:

Very definite fibrous structure. Cannot be pulverized in a mortar, but must be ground in a mill. Swells in pure water and if a sufficient amount of water is present, forms a cloudy, viscous solution after several days. Similar highly viscous, colloidal systems are also formed with NH<sub>4</sub> salts, with highly diluted Ca or Mg salt solutions and with LiCl but not, however, with K ions. Can be precipitated with NaCl solution, very concentrated NH<sub>4</sub>-KCl solution and with alcohol. M.p. 630-650°C. d. 2.56 to 2.62. The anions of the salt consist of very long PO<sub>3</sub> chains, and the Na ion can be replaced reversibly with other cations, as in ion exchange. Thilo considers Kurrol's salt to be the crystalline form of Graham's salt. It appears in two forms a and b, of which form a, obtained from Graham's salt by annealing, transforms into form b on standing, on treatment with water or through purely mechanical stresses. The two forms show different powder patterns. They also differ by the fact that form a is quantitatively transformed into Madrell's salt on annealing between 420 and 490°C, while form b transforms quantitatively into trimetaphosphate between 390 and 600°C.

#### REFERENCES:

- G. Tammann. J. prakt. Chem. [2] <u>45</u>, 467 (1892).
- H. Huber and K. Klumpner. Z. anorg. allg. Chem. 251, 213 (1943).
- P. Pascal. Compt. Rend. Hebd. Séances Acad. Sci. <u>178</u>, 211 and 1541 (1924);
- E. Thilo. Angew. Chem. <u>63</u>, 511 (1951).

## **Metaphosphates**

#### SODIUM TRIMETAPHOSPHATE, Na<sub>3</sub>P<sub>3</sub>O<sub>9</sub>·6H<sub>2</sub>O

Sodium trimetaphosphate, discovered by Fleitmann and Henneberg, is best prepared, according to Von Knorre, by heating a mixture of Na<sub>2</sub>HPO<sub>4</sub> with NH<sub>4</sub>NO<sub>3</sub>. According to Karbe and Jander, a temperature of  $310-320^{\circ}$ C is most favorable:

$$\begin{array}{r} 3\,\mathrm{Na_{2}HPO_{4} \cdot 12}\,\mathrm{H_{2}O}\,+\,3\,\mathrm{NH_{4}NO_{3}}\,=\,\mathrm{Na_{3}P_{3}O_{9}}\,+\,3\,\mathrm{NaNO_{3}}\,+\,3\,\mathrm{NH_{3}}\\ 1074.6\,\,\,\,240.0\,\,\,\,306.0\,\,\,\,255.0\,\,\,\,51.0\,\,\,\,\\ +\,\,39\,\mathrm{H_{2}O}\\ 702.6\,\,\,\,\end{array}$$

The salt may also be formed by annealing Graham's salt (see above) at  $520^{\circ}$ C.

I. A mixture of 60 g. of  $Na_2HPO_4$  and 17 g. of  $NH_4NO_3$  is heated for six hours at 310-320°C. The mixing produces a very sharp cooling of the salt mass. The reaction product is white and crystalline. It is dissolved in cold water, filtered and allowed to crystallize. (According to Tammann, two layers are formed on leaching, of which the lower contains Graham's salt, while the trimetaphosphate can be crystallized out of the upper one after layer separation.)

II. Graham's salt is heated for 12 hours at  $520^{\circ}$ C. The absence of precipitation with Ba, Ag, Pb or other salts of heavy metals indicates the end point of the reversible reaction. To prepare the hexahydrate, 51 g. of anhydrous salt is dissolved in 160 ml. of water at room temperature and 45 ml. of saturated NaCl solution is added. After stirring for four hours the crystals are filtered by suction and air dried. The yield is 23 g. (33%) of hexahydrate.

#### PROPERTIES:

Formula weight 413.99. Triclinic prisms. d 2.476. Nonhygroscopic; loses its water of crystallization over  $H_2SO_4$  and at 100°C. On extended storage at above 20°C, loses water and finally transforms to the anhydrous salt. Solubility: 4.5 parts  $H_2O$  per part of salt; insoluble in alcohol. The transformation to orthophosphate requires repeated evaporation with mineral acids. Trimetaphosphate forms no precipitates with Ba, Ag or Pb salts. However, the salt obtained in the above procedures gives a slight opalescence with silver nitrate. **REFERENCES:** 

- T. Fleitmann and W. Henneberg. Liebigs Ann. Chem. <u>65</u>, 304 (1848).
- G. von Knorre. Z. anorg. allg. Chem. 24, 381 (1900).
- K. Karbe and G. Jander. Kolloid-Beihefte 54, 35 and 36 (1942).

SODIUM TETRAMETAPHOSPHATE,  $Na_4P_4O_{12} \cdot n H_2O$  (n = 10 and 4)

The starting material is  $Cu_2P_4O_{12}$ , prepared by heating a mixture of CuO and  $H_3PO_4$ . This is then reacted with  $Na_2S$  to yield the sodium salt (Andress, Gehring and Fischer):

 $\begin{array}{rrrr} {\rm Cu}_2{\rm P}_4{\rm O}_{12} \ + \ 2\,{\rm Na}_2{\rm S} \ = \ {\rm Na}_4{\rm P}_4{\rm O}_{12} \ + \ 2\,{\rm Cu}{\rm S} \\ {\rm (9\,H_2O)} & {\rm (4\,H_2O)} \\ {\rm 443.0} & {\rm 480.4} & {\rm 479.9} & {\rm 191.2} \end{array}$ 

Freshly precipitated CuO is added slowly and in small portions to a 5% excess of 76.9%  $H_3PO_4$ . This yields a paste, which becomes bright blue on standing overnight. To drive out the free water, the mass is heated in a porcelain evaporating dish over a Bunsen burner. At first, the heating is gentle; then the material is heated for several hours at temperatures not exceeding 430°C. The fine crystals formed are washed eight times with hot water, until the wash water is neutral.

A 72-g. portion of this  $Cu_2P_4O_{12}$  is added in small portions to a vigorously stirred solution of 78 g. of  $Na_2S \cdot 9H_2O$  in 750 ml. of  $O_2$ -free water. The filtered solution is concentrated to 1/3 its volume, filtered and precipitated with alcohol or acetone. After recrystallization, pure white crystals are obtained. The yield is 52 g., or 67% of theoretical. A further fraction may be obtained from the mother liquor (Thilo and Rätz).

Other preparative methods: Free tetrametaphosphoric acid,  $H_4P_4O_{12}$ , is formed in low-temperature hydration of  $P_4O_{10}$  (form M, see p. 541) (Bell, Audrieth and Hill). The Na salt is readily obtained from this solution by neutralization with NaOH.

With vigorous stirring, 50 g. of  $P_4O_{10}$  is slowly added to 300 ml. of water. The temperature should never exceed 15°C. When the  $P_4O_{10}$  is completely dissolved, the solution is neutralized with 30% NaOH to a pH of 7 (about 98 g. of NaOH solution is needed). Then 30 g. NaCl is added and the mixture allowed to stand overnight. The decahydrate is formed below 25°C, while the tetrahydrate is obtained above 40°C. The crystals are filtered off, washed with water at 5-10°C, and air-dried. The yield is 60-65%.

According to Such and Tomlinson, the tetrahydrate may be obtained by treating  $P_4O_{10}$  with  $Na_2CO_3 \cdot 10H_2O$  or with a cold suspension of NaHCO<sub>3</sub>. The yield is 50%.

Purification of  $Na_4P_4O_{12}$  by ion exchange is described by Barney and Gryder.

**PROPERTIES:** 

 $Na_4P_4O_{12} \cdot 4H_2O$  formula weight 479.93. The decahydrate loses six molecules of water on heating to 40°C. The tetrahydrate occurs in two polymorphic forms with the transition point at 54°C. The transition is irreversible. The anhydride is formed at 100°C. It transforms into trimetaphosphate at about 400°C. The tetraphosphate (p. 548) is formed on careful hydrolysis with NaOH; at 100°C the ortho- and triphosphates are formed:  $P_4O_{12}^{4-} + 4OH^- = P_3O_{10}^{5-} + PO_4^{3-} + 2H_2O$ .

**REFERENCES:** 

- K. R. Andress, W. Gehring and K. Fischer. Z. anorg. allg. Chem. 260, 331 (1949).
- E. Thilo and R. Rätz. Z. anorg. allg. Chem. 260, 255 (1949).
- R. N. Bell, L. F. Audrieth and O. F. Hill. Ind. Eng. Chem. <u>44</u>, 568 (1952).
- J. E. Such and R. H. Tomlinson, as cited by B. Topley. Quart. Revs. 3, 345 (1949).
- D. L. Barney and J. W. Gryder. J. Amer. Chem. Soc. <u>77</u>, 3195 (1955).

## Orthophosphorous Acid

#### H<sub>3</sub>PO<sub>3</sub>

Orthophosphorous acid is prepared by the hydrolysis of PCl<sub>3</sub>:

$$\begin{array}{r} PCl_{3} + 3 H_{2}O = H_{3}PO_{3} + 3 HCl \\ 137.3 & 54.0 & 82.0 & 109.4 \end{array}$$

In order to moderate the violent reaction, Milobedzki and Friedmann recommend using concentrated hydrochloric acid. In this procedure two layers are formed, so that only a part of the  $PCl_3$  enters into the reaction at any time. The product solution is concentrated. Simon and Feher heat the commercial acid for twelve hours at 80°C and allow it to cool over  $P_2O_5$ . In this way they obtain a crystalline acid containing 99.3%  $H_3PO_{13}$ .

#### **PROPERTIES:**

White crystalline mass; deliquesces on standing in air; very soluble in water. M.p. 74°C. The solid acid disproportionates on heating:  $4 H_3PO_3 = PH_3 + 3H_3PO_4$ . d (21°C) 1.65.

REFERENCES:

- T. Milobedzki and M. Friedmann. Chemik Polski <u>15</u>, 76 (1917), as cited by Chem. Zentr. 1918 I, 933.
- A. Simon and F. Feher. Z. anorg. allg. Chem. 230, 298 (1937).

## Hypophosphorous Acid

## H<sub>3</sub>PO<sub>2</sub>

According to Klement, free, crystalline hypophosphorous acid can be obtained in a highly purified form in a simple procedure consisting of treating  $Na_2H_2PO_2$  with the H<sup>+</sup> form of a cation exchanger (I). According to Simon and Feher, one can also start directly from a commercial solution of  $H_3PO_2$  (about 50%, d 1.274), which is then concentrated. In the Thomsen preparation (II),  $Ba(H_2PO_2)_2$  (see p. 557) is decomposed with the stoichiometric quantity of  $H_2SO_4$ :

 $\begin{array}{ccc} Ba(H_2PO_2)_2 \cdot H_2O \,+\, H_2SO_4 \,=\, 2\,H_3PO_2 \,+\, BaSO_4 \,+\, H_2O\\ 285.4 & 98.1 & 132.0 & 233.4 & 18.0 \end{array}$ 

A procedure for the purification of the acid is given by Jenkins and Jones (III).

I. About 70 g. of commercial cation exchange resin is placed over a cotton wad in a glass column with a bulb and a drain stopcock. The tube I.D. is 25 mm, and its length is about 25 cm. The tube is completely filled with water. The resin is left to swell for a few hours and then, by opening the stopcock, 5 NHCl is drawn from the bulb into the layer covering the resin until the resin is entirely covered with acid. After 15 minutes the acid is drawn off and distilled water is repeatedly added until the wash water is acid free. The resin layer is completely drained and a solution of 15 g. of NaH<sub>2</sub>PO<sub>2</sub> in 60 ml. of water is added to it. Any air bubbles present in the resin are removed by shaking. After 15 minutes of treatment, the solution is drained drop-by-drop and the resin is rinsed first with 50 ml., and then with 25 ml. of distilled water. The washings are combined with the solution and the combined solution of free H<sub>3</sub>PO<sub>2</sub> is then filtered and evaporated on a water bath. Further dehydration proceeds via the method of Simon and Feher. The acid is placed in high vacuum over  $P_2O_z$  and kept there until the P<sub>2</sub>O<sub>5</sub> begins to turn red-brown because of the reaction with the volatile acid. The acid is then crystallized in a cold bath. The crystals are freed of mother liquor on cooled clay plates, and are melted and allowed to recrystallize by freezing. The product is about 98% pure.

The ion-exchange resin is again washed twice with distilled water, using 50 ml. each time; it is reactivated with 5N HCl, and, after thorough washing until the eluent is acid-free, it can be used again.

It is recommended that a batch of at least 60 g. of  $NaH_2PO_2$  be processed (in four portions) at one time. This procedure has the advantage that a pure solution of the free acid is obtained directly, without having to filter off the finely divided  $BaSO_4$  as in method II. Evaporation in high vacuum assures that no decomposition of  $H_3PO_2$  will occur and that the product will therefore be free of traces of  $H_3PO_4$ .

II. A solution of 285 g, of  $Ba(H_2PO_2)_2$  in five liters of water is decomposed with a solution of 100 g. of concentrated H<sub>2</sub>SO<sub>2</sub> in 3-4 times its weight of water. The mixture is well stirred and allowed to stand for a day; the BaSO<sub>4</sub> settles out. The supernatant liquid is siphoned off. The product solution of H<sub>3</sub>PO<sub>2</sub>, which, if proper amounts of reactants are used, contains only traces of Ba, is evaporated by boiling in a porcelain dish until concentrated to about 1/10 of the original volume. It is then evaporated in a Pt dish while stirring with a thermometer until the temperature reaches 105°C. The bulb of the thermometer must be completely immersed without, however, touching the bottom of the dish. It is therefore impractical to work with less than the amount given above, since otherwise the volume of the concentrated acid is too small. The Pt dish is heated with a burner covered with a wire screen, so that the heat is evenly distributed at the bottom of the vessel.

When the temperature has risen to  $105^{\circ}$ C, the liquid is quickly filtered and the colorless filtrate is concentrated until the temperature rises to  $110^{\circ}$ C. It should not be allowed to boil. The temperature is now kept constant for 15 minutes and is then gradually raised to  $130^{\circ}$ C, again avoiding boiling. The acid simmers gently, shows no gas bubbles, and has no odor of PH<sub>3</sub>. However, some vapor is present because traces of it volatilize. By carefully heating, the temperature can be raised even to  $138^{\circ}$ C without decomposition. After the acid has been heated for about 10 minutes at  $130^{\circ}$ C, the flame is removed and the liquid is cooled and filtered into a glass-stoppered flask.

The glass vessel is now cooled down to a few degrees below zero, and if crystallization does start, the bottom of the vessel is scratched with a glass rod and the material allowed to stand. III. Jenkins and Jones recommend the following procedure for the purification of the acid. The directions must be followed most carefully. Commercial 50% acid (600 ml.) is placed in a one-liter suction flask, provided with a two-hole stopper for a thermometer and a gas inlet tube with a coarse fritted glass end. The suction flask is placed on a hot plate and connected to an aspirator.

A fast stream of N<sub>2</sub> is introduced until all air is displaced. The N<sub>2</sub> stream is throttled down to a few bubbles, and the aspirator and the hot plate are turned on. Evaporation to 300 ml. is carried out at 40°C. The temperature should not exceed 45°C. The cooled liquid is poured into a wide-neck Erlenmeyer flask, which is then well stoppered and placed in a Dry Ice-acetone bath. After crystallization, which sometimes must be started by scratching the walls, the flask is allowed to stand for 12 hours at about 5°C. At that point the liquid should constitute 30-40% of the flask contents. Further operations must be carried out in a refrigerated space. The crystals are quickly filtered through filter paper and the filtrate is discarded. The crystals are pressed dry and placed in a crystallizing dish, in which they are allowed to stand until about 20-30% of the contents liquefy. The residue is filtered off and the process is repeated. The almost pure residue (about 10% yield) is stored over  $Mg(ClO_4)_2$  in a vacuum desiccator in a refrigerated room. To obtain large crystals, the material can be recrystallized from n-butanol.

PROPERTIES:

Anhydrous  $H_3PO_2$  crystallizes in colorless lamellae which readily dissolve in water. M.p. 26.5°C. Heating the anhydrous acid to 130-140°C results in disproportionation to PH<sub>3</sub> and H<sub>3</sub>PO<sub>3</sub>. These decompose further into PH<sub>3</sub> and H<sub>3</sub>PO<sub>4</sub>. d (19°C) 1.493.

REFERENCES:

J. Thomsen. Ber. dtsch. chem. Ges. 7, 994 (1874).

R. Klement. Z. anorg. allg. Chem. 260, 267 (1949).

A. Simon and F. Fehér. Z. anorg. allg. Chem. 230, 298 (1937).
 W. A. Jenkins and R. T. Jones. J. Amer. Chem. Soc. 74, 1353 (1952).

# **Barium Hypophosphite**

## $Ba(H_2PO_2)_2 \cdot H_2O$

Barium hypophosphite and  $PH_3$  are produced in the reaction of white phosphorus with  $Ba(OH)_2$ :

A solution of 120 g. of crystalline  $Ba(OH)_2$  in 1200 ml. of water is heated for about four hours with 30 g. of white P in a roundbottom flask. The flask is provided with a long glass tube, extending to the stack of the exhaust hood, to conduct away the spontaneously igniting phosphine mixture. When the P is completely dissolved,  $CO_2$  is introduced to precipitate the excess  $Ba(OH)_2$ . The precipitate is filtered off and washed with hot water. The solution and the wash water are combined and evaporated to half the original volume, refiltered, and evaporated until crystallization begins. Some alcohol is then added and the mixture is left to cool. The resulting crystals are suction-filtered and the mother liquor again evaporated until crystallization takes place. The accumulated salts are combined and recrystallized from hot water. The yield is 40-60 g.

**PROPERTIES:** 

Formula weight  $Ba(H_2PO_2)_2$ , 267.34;  $Ba(H_2PO_2)_2 \cdot H_2O$ , 285.36. Colorless, tabular prisms (monoclinic) with a pearly sheen. Insoluble in alcohol, readily soluble in water. Solubility: 28.6 g. per 100 g. cold water; 33.3 g. per 100 g. boiling water. d (17°C) 2.90.

**REFERENCE:** 

H. Rose. Pogg. Ann. 9, 370 (1827).

# Hypophosphoric Acid

## H<sub>4</sub>P<sub>2</sub>O<sub>6</sub>

According to Salzer, free crystalline hypophosphoric acid is obtained when lead hypophosphate, prepared from disodium dihydrogen hypophosphate (see below), is decomposed with  $H_2S$  and the resulting acid solution evaporated (Baudler):

 $\begin{array}{c} \mathrm{Na_{2}H_{2}P_{2}O_{6}} + 2\ \mathrm{Pb}(\mathrm{CH_{3}CO_{2}})_{2} = \mathrm{Pb_{2}P_{2}O_{6}} + 2\ \mathrm{Na}(\mathrm{CH_{3}CO_{2}}) + 2\ \mathrm{CH_{3}CO_{2}H} \\ (6\ \mathrm{H_{3}O}) \\ 314.0 \\ 758.7 \\ 572.4 \\ 164.1 \\ 120.1 \\ \mathrm{Pb_{2}P_{2}O_{6}} + 2\ \mathrm{H_{2}S} = \mathrm{H_{4}P_{2}O_{6}} + 2\ \mathrm{PbS} \\ 572.4 \\ 68.2 \\ 162.0 \\ 478.6 \end{array}$ 

A clear solution of 425 g. of  $Pb(C_2H_3O_2)_2 \cdot 3H_2O$  in 850 ml. H<sub>2</sub>O (if necessary, treated with several drops of glacial acetic acid and filtered) is stirred into a hot solution of 174 g. of Na<sub>2</sub>H<sub>2</sub>P<sub>2</sub>O<sub>6</sub>  $\cdot$  6H<sub>2</sub>O (twice recrystallized) in one liter of water. A copious precipitate is formed immediately and is allowed to settle by standing overnight; the supernatant liquid is then decanted; the precipitate is filtered on a suction filter and washed carefully with small amounts of H<sub>2</sub>O. It is then placed in a wide-mouth



Fig. 197. Evaporation apparatus for the preparation of hypophosphoric acid. a) Drain; b) vaporizer (actually five heating tubes instead of the three shown); h) heating medium; k) cooling liquid; l) air intake;  $s_1, s_2$ ) groundglass ball and socket joints; v) vacuum connection.

flask with 500 ml. of H<sub>a</sub>O, mechanically agitated for several hours, and suction-filtered. This is repeated. The pure  $Pb_2P_2O_6$ , dried as much as possible by suction, is then suspended as evenly as possible, using vigorous stirring, in one liter of doubledistilled water placed in a threeliter, large diameter cylindrical vesfast stream of sel. Α H<sub>2</sub>S is bubbled through for about seven hours, while the vessel is cooled with ice. The HaS flow is shut off when the precipitate has taken on a uniform jet-black color. If the slurry becomes too thick, more H<sub>2</sub>O must be added. The precipitate of PbS is suction-filtered and freed as completely as possible from the associated residual acid solution by careful washing with some H<sub>2</sub>O. Dissolved H<sub>2</sub>S is removed in a stream of N<sub>a</sub>. After repeated filtration through activated charcoal, using a fritted glass filter, the solutionabout one liter containing about 7% $H_4P_2O_6$ —is placed in the reservoir of a natural circulation evaporator (see Fig. 197). The capacity of about the vaporizer is 60 ml.: and with the amounts specified here, crystallization of the  $H_4P_2O_6 \cdot 2H_2O_6$ occurs shortly before this volume is attained. The procedure is carried out at a bath temperature of 65°C and a pressure of 3-7 mm.

The condenser and the receiver are cooled with methanol to -10 to  $-15^{\circ}$ C. Except for the ground-glass ball and socket joint  $s_1$ , which is coated with grease, all other joints are sealed with purified molten paraffin. After evaporating for about 2 hours, the saturated  $H_2P_2O_6 \cdot 2H_2O$  solution in the reservoir is drawn back by suction into the reservoir or drained through  $\alpha$ . Following cooling at  $-78^{\circ}$ C and suction filtering through a fritted glass filter (filtration protected from atmospheric moisture), 40-43 g. of crystalline  $H_4P_2O_6 \cdot 2H_2O$  contaminated with 0.3-1.0%  $H_3PO_3$  is obtained. The mother liquor is contaminated by large amounts of hydrolysis products.

When purity requirements are lower, the acid solution is simply filtered to remove the PbS and evaporated by boiling. This can be continued as long as it still contains a large amount of water. In order to concentrate it to a syrup, it is evaporated in vacuum over  $CaCl_2$ . The  $H_4P_2O_6 \cdot 2H_2O$  precipitates upon cooling of the syrupy liquid.

PROPERTIES:

Formula weight  $(H_4P_2O_6 \cdot 2H_2O)$  198.01. Large rhombic plates, hygroscopic, very readily soluble in water. M.p. (corr.) 62°C. When placed in vacuum over  $P_2O_5$  for two months it forms the anhydrous acid which has a broad melting range beginning at about 73°C. The anhydrous acid liquefies at room temperature within six days but remains unchanged for long periods at 0-5°C if  $H_2O$  is excluded; the same holds for the dihydrate. An aqueous solution of the acid is stable; it decomposes on boiling with sulfuric acid to form  $H_3PO_3$  and  $H_3PO_4$ . The use of rubber gloves and protective glasses is strongly recommended when working with the solid acid or concentrated solutions. Contact with the skin produces blisters which heal very slowly; damage to the cornea results on contact with the eyes.

**REFERENCES:** 

T. Salzer. Liebigs Ann. Chem. <u>187</u>, 322 (1877); <u>211</u>, (1882).
M. Baudler. Z. anorg. allg. Chem. <u>279</u>, 115 (1955).
H. Remy and H. Falius. Naturwiss. <u>43</u>, 177 (1956).

# Disodium Dihydrogen Hypophosphate

## $Na_2H_2P_2O_6 \cdot 6H_2O$

Leininger and Chulski have demonstrated that disodium dihydrogen hypophosphate is produced in the oxidation of red P with NaClO<sub>2</sub> in a yield of about 42%. In addition, 19% orthophosphate, 35% phosphite, and 2% hypophosphite are formed. The procedure, somewhat cumbersome because of the large cooling apparatus required, has been improved and simplified by Baudler and, later, by Remy and Falius. According to the latter authors, special apparatus is unnecessary. Sufficient phosphorus must be available, however, and the heat of reaction must be rapidly removed. The yield is better than that achieved by Leininger and Chulski.

Red phosphorus (100 g.) is suspended in one liter of water placed in an open, round-bottom flask, which should be well cooled by running water over it. Over a period of five hours, a solution of 170 g. of NaClO<sub>2</sub> in 350 ml. of H<sub>2</sub>O is added in drops with vigorous stirring. The unreacted P is then filtered. Following the addition of 10-12 g. of activated charcoal, the mixture is allowed to stand overnight. The activated charcoal is then filtered, and the clear solution is treated with a 15% solution of sodium hydroxide to give a pH of 5.4 (measured with a glass electrode). At this point, cationic impurities precipitate out. To complete the precipitation, the mixture is boiled, filtered, and placed overnight in a refrigerator. The precipitated crystals are suction-filtered, washed with ice water, and dried in air. The yield is about 120 g. Considerably larger quantities of material can be reacted if sufficient cooling is provided. The salt is recrystal-lized once or more from water for purification.

#### PROPERTIES:

Formula weight  $(Na_2H_2P_2O_6 \cdot 6H_2O)$  314.04. Forms tabular, monoclinic crystals. Stable in air at room temperature; loses its water of crystallization upon gentle heating. The salt, dehydrated at 100°C, melts at 250°C and decomposes at red heat, evolving flammable phosphine. d (20°C) 1.8.

**REFERENCES:** 

E. Leininger and T. Chulski. J. Amer. Chem. Soc. <u>71</u> 2385 (1949).
M. Baudler. Z. anorg. allg. Chem. <u>279</u>, 115 (1955).
H. Remy and H. Falius. Naturwiss. <u>43</u>, 177 (1956).

# Tetrasodium Hypophosphate

# $Na_4P_2O_6\cdot 10~H_2O$

According to Salzer, tetrasodium hypophosphate is obtained by neutralizing a solution of  $Na_2H_2P_2O_6$  with  $Na_2CO_3$ :

 $\begin{array}{rrr} Na_{2}H_{2}P_{2}O_{6} \cdot 6 H_{2}O + Na_{2}CO_{3} + 3 H_{2}O = Na_{4}P_{2}O_{6} \cdot 10 H_{2}O + CO_{2} \\ 314.1 & 106.0 & 54.0 & 430.1 & 44.0 \end{array}$ 

A solution of  $Na_2H_2P_2O_6$  in 50 parts of water is treated with an excess of concentrated  $Na_2CO_3$  solution. The neutral salt precipitates in small, snowflake-like crystals. If a hot concentrated solution of the disodium salt is treated with  $Na_2CO_3$ solution, the entire solution solidifies into silky needles of the tetrasodium salt.

PROPERTIES:

Glossy crystalline needles. Recrystallizable from water without change. The aqueous solution gives an alkaline reaction with phenolphthalein. d 1.823. **REFERENCE:** 

T. Salzer. Liebigs Ann. Chem. 194, 29 (1878).

# Barium Dihydrogen Hypophosphite BaH<sub>2</sub>P<sub>2</sub>O<sub>6</sub>·2 H<sub>2</sub>O

Obtained by precipitation of a solution of  $Na_2H_2P_2O_6$  with  $BaCl_2$  solution.

 $\begin{array}{rl} Na_2H_2P_2O_6 + BaCl_2 = BaH_2P_2O_6 + 2 NaCl \\ (6 H_2O) & (2 H_2O) & (2 H_2O) \\ 314.1 & 244.3 & 333.4 & 116.9 \end{array}$ 

A solution of 144 g. of  $Na_2H_2P_2O_6 \cdot 6 H_2O$  (twice recrystallized) is prepared with heating in three liters of water, and 12 ml. of hydrochloric acid (d 1.12) is added. A hot, filtered solution of 112 g. of  $BaCl_2 \cdot 2 H_2O$  in 720 ml. of water is gradually stirred in. The initially gelatinous precipitate becomes coarsely crystalline after standing overnight. It is suction-filtered on a fritted glass filter, washed with ice water, and dried on a clay plate. The yield is about 140 g.

PROPERTIES:

Clear, very hard, needle-shaped monoclinic crystals. Very slightly soluble in water. Loses its water of crystallization at  $140^{\circ}$ C.

**REFERENCES:** 

C. Bensa. Z. anorg. allg. Chem. <u>6</u>, 132 (1894);
M. Baudler. Z. anorg. allg. Chem. <u>279</u>, 115 (1955).

## **Potassium Peroxydiphosphate**

## $K_4P_2O_8$

According to Fichter et al., potassium peroxydiphosphate is formed during the anodic oxidation of  $KH_2PO_4$ , in the presence of fluoride and chromate, through the decomposition of the intermediate peroxyphosphoric acid:

 $3 H_3 PO_5 = H_4 P_2 O_8 + H_3 PO_4 + H_2 O_2 + O_2$ 

A solution of 302.2 g. of  $KH_2PO_4$ , 198 g. of KOH, 120 g. of KF and 0.355 g. of  $K_2CrO_4$  in one liter of water is prepared. A 215-ml. portion of this solution is electrolyzed in a large platinum dish (which serves as the anode), with a rapidly rotating bent Pt wire used as the cathode. The anode current density is 0.02-0.03 amp./cm<sup>?</sup> The electrolyte temperature is maintained below 14°C by external cooling. The current is interrupted after three hours and the material is allowed to stand overnight at room temperature. During this period, the intermediate  $H_3PO_5$  decomposes, evolving  $O_2$  and re-forming orthophosphate. A certain amount of  $K_4P_2O_8$  is also formed (see above equation), while the already existing  $K_4P_2O_8$  remains unchanged. Following a second electrolysis for two hours, the material is again allowed to stand overnight. Finally, the solution is electrolyzed a third time for one hour and again permitted to stand overnight.

The solution is then evaporated on a water bath, with stirring, while a stream of air is directed across the upper surface so that its temperature never rises above  $80^{\circ}$ C. By recrystallizing three times, 96.4-99.8% pure K<sub>4</sub>P<sub>2</sub>O<sub>8</sub> is obtained. The yield is about 80%.

PROPERTIES:

Formula weight 346.35. Very gradually evolves free I2 with KI.

REFERENCES:

F. Fichter and A. Rins y Miro. Helv. Chim. Acta 2, 3 (1919).

F. Fichter and E. Gutzwiller. Helv. Chim. Acta 11, 323 (1928).

# Phosphorus Trisulfide

## $P_4S_3$

Phosphorus trisulfide can be synthesized either by Stock's method (I), which involves fusing red P and S and carefully purifying the crude product by recrystallization from  $CS_2$ , followed by a second recrystallization from benzene, or according to German patent 309,618 (F. C. Frary), which uses white P in a high-melting inert solvent (II).

 $\begin{array}{l} 4\,P\,+\,3\,S\,=\,P_4S_3 \\ 123.9 \quad 96.2 \quad 220.1 \end{array}$ 

I. STOCK'S METHOD

The starting materials and the solvents must be carefully purified prior to use. Very pure crystalline S is powdered as finely as possible and dried at  $100^{\circ}$ C. Red P is purified by the method given on p. 519. Carbon disulfide is shaken for 24 hours with CaCl<sub>2</sub> and Hg, distilled and stored in brown bottles over P<sub>2</sub>O<sub>5</sub>. Benzene is distilled and stored over P<sub>2</sub>O<sub>5</sub>.

An intimate mixture of 155 g. of red P and 96 g. of S is prepared and portions (40-50 g.) are reacted in a large diameter open test tube under a stream of dry  $CO_2$ . It is recommended that a pan filled with sand be placed under the test tube since the molten mass will immediately catch fire if the tube breaks. The test tube is first lightly preheated over its entire length to about 100°C and is then heated with a small flame in one spot at the upper edge of the mixture until the onset of the reaction. As soon as the entire charge has reacted (thereby becoming molten), it is heated to the point where distillation begins and then cooled in a stream of  $CO_2$ . The cake is then pulverized and extracted with hot  $CS_2$  (200 g. for each 100 g. of material). By evaporation of the  $CS_2$ , an almost theoretical yield of crude sulfide, which melts between 130 and 150°C, is obtained.

Purification is carried out by adding hot water to the powdered product (200 ml. for each 100 g. of material) in a one-liter roundbottom flask; a strong jet stream is introduced into the mixture for one hour so that it is continually agitated. On cooling, the supernatant liquid is decanted as completely as possible and the  $P_4S_3$  is extracted in a separatory funnel with about 150 g. of  $CS_2$ for each 100 g. of crude sulfide. The solution, which is easily separated from the aqueous layer and the undissolved impurities, is shaken for 12 hours with  $P_2O_5$ ; it is then evaporated on a steam bath until crystals form and then evaporated in aspirator vacuum until dry. The final drying may not be done over a steam bath because the sulfide is partially decomposed at this temperature. A 100-g. quantity of crude product yields about 98 g. of compound, m.p. 169-171.5°C, which still contains some CS<sub>2</sub>. This is removed by recrystallization from benzene in the following manner: A reflux condenser is connected with a ground-glass joint to the 5-cm, -diameter neck of a 750-ml. round-bottom flask; an extraction cell is suspended from the lower end of the condenser by a wire. The cell consists of a piece of glass tubing 10 cm. long and 4 cm. in diameter which has been somewhat narrowed at the lower end and fitted with several small hooks at the upper. Hard filter paper is secured over the lower opening. The cell holds about 80-100 g. of the compound to be recrystallized. A 300-ml. portion of benzene is kept at a vigorous boil in the surrounding flask. In six hours, about 100 g of  $P_4S_3$  is extracted from the cell. The small, uniform crystals which precipitate out of the benzene solution are carefully suction-filtered, with care not to cool too strongly and protection from atmospheric moisture. It is freed of residual solvent by passing dry H<sub>2</sub> over it. This is

conveniently done in an upright glass tube, not too wide, and is continued so long as the  $H_2$  gives off an odor of benzene. Additional, less pure product (5-7% of the crude) can be recovered by concentrating the benzene solution. The yield of pure  $P_4S_3$  is about 92% of the crude product.

#### II. PROCEDURE ACCORDING TO GERMAN PATENT 309, 618

White P is dissolved in (virtually nonflammable)  $\alpha$ -chloronaphthalene and the stoichiometric quantity of S is added. The solvent acts as a diluent controlling the rate of reaction and causes crystallization of the sulfide. When the reaction is complete, the mixture is cooled with stirring. Most of the P<sub>4</sub>S<sub>3</sub> precipitates as a fine powder.

PROPERTIES:

Yellowish-green, long rhombic needles; stable in the air. M.p. 172.5°C, b.p. 407°C;  $d_4^{2\circ}$  2.03. In the absence of O<sub>2</sub> and moisture, remains stable above 700°C. Decomposed by water at elevated temperatures, evolving H<sub>2</sub>S. Soluble in CS<sub>2</sub> and benzene. These solutions become turbid in air almost immediately and gradually deposit a copious, yellowish-white precipitate.

REFERENCES:

- I. A. Stock. Ber. dtsch. chem. Ges. 43, 150 (1910).
- II. F. C. Frary. German Patent 309,618, Class 12i, December 2, 1918, cited in Chem. Zentr. 1919 II, 55.

# **Phosphorus Pentasulfide**

## $P_4S_5$

According to Boulouch, as confirmed by Treadwell and Beeli, phosphorus pentasulfide is easily prepared by exposing a solution of  $P_4S_3$  and S in  $CS_2$  to light in the presence of a small amount of iodine as catalyst.

$$P_4S_3 + 2S = P_4S_5$$
  
220.1 64.1 284.2

A solution of 22 g. of  $P_4S_3$  and 7 g. of S in 200 ml. of  $CS_2$  is prepared, and 0.2 g. of iodine is added. The mixture is exposed to light for one or two days (for three days in diffuse daylight at room temperature, according to Treadwell and Beeli). The crystals that precipitate are washed with  $CS_2$ , dried, powdered, washed again and dried in air. The crystals usually contain residual  $CS_2$ , which can be removed by melting. The yield is 23 g.

PROPERTIES:

Sulfur-yellow crystals which, when slowly heated, melt between  $170^{\circ}$  and  $220^{\circ}$ C. d (25°C) 2.17.

REFERENCES:

R. Boulouch. Compt. Rend. Hebd. Séances Acad. Sci. 138, 363 (1904); W. D. Treadwell and C. Beeli. Helv. Chim. Acta 18, 1161 (1935).

## **Phosphorus Heptasulfide**

 $P_4S_7$ 

According to Stock, phosphorus heptasulfide is produced in a manner analogous to that for  $P_4S_3$  (see p. 563).

 $4 P + 7 S = P_4 S_7$ 123.9 224.5 348.4

The starting materials and the solvent must be purified as specified on p. 563.

An intimate mixture of 100 g. of red P and 173 g. of S is prepared and reacted in portions (40-50 g.), after which it is heated until pronounced distillation occurs. The cooled produce is finely powdered and recrystallized from  $CS_2$  in the extraction apparatus described on p. 564. The extraction cell is charged with 40-50 g. of crude product; the flask is charged with about 300 ml. of  $CS_2$ , which must be maintained at a vigorous boil. Because of the low solubility of  $P_4S_7$ , the extraction requires about 48 hours. The sulfide, which precipitates as glittering crystals, is then recrystallized in the same manner. The crystals are suction-filtered, care being taken to exclude atmospheric moisture and not to cool too much. It is dried in a stream of hydrogen at 100°C.

PROPERTIES:

Almost colorless, faintly yellow monoclinic prisms. M.p.  $310^{\circ}$ C, b.p. (760 mm.) 523°C. The best solvent is CS<sub>2</sub>. One part of P<sub>4</sub>S<sub>7</sub> dissolves in 3500 parts of CS<sub>2</sub> at 17°C and in 20,000 parts at 0°C. The very slight solubility in CS<sub>2</sub> distinguishes P<sub>4</sub>S<sub>7</sub> from the other phosphorus sulfides. Slowly decomposed by cold water, rapidly by hot water, forming  $H_2S$ . d (17°C) 2.19.

REFERENCE

A. Stock. Ber. dtsch. chem. Ges. 43, 414 (1910).

## **Diphosphorus Pentasulfide**

## $P_2S_5$

According to Stock and Herscovici, the preparation of diphosphorus pentasulfide is carried out by the same synthesis as for  $P_4S_3$  (p. 563). The reagents must also be purified according to the instructions given there.

 $\begin{array}{l} 2 P + 5 S = P_2 S_5 \\ 62.0 \quad 160.3 \quad 222.3 \end{array}$ 

An intimate mixture of 100 g. of red P and 260 g. of S (2 g. atoms of P to 5 g. atoms of S, with a 1% excess of S) is reacted in portions and heated until vaporization begins. The cooled product is coarsely ground and placed in a Pyrex tube. The tube is evacuated, using an aspirator, fused to seal it and heated for several hours at about 700°C in an iron tube packed with sand and closed with a screw cap. After cooling, the glass tube is broken; the contents are pulverized and recrystallized from boiling CS<sub>2</sub> in the extraction apparatus (about 300 ml. per 50 g. of sulfide). The precipitated sulfide is twice recrystallized in the same manner, and the crystals are dried at 100°C in a stream of H<sub>2</sub>. If the crude product is not heated under pressure, a pure product can only be obtained by repeated recrystallization. The yield is 60%.

PROPERTIES:

Light yellow, almost colorless triclinic crystals with a  $P_4S_{10}$  molecular lattice (Vos and Wiebenga). M.p. 276°C, b.p. 514°C;  $d_4^{20}$  2.03. Soluble with difficulty in boiling CS<sub>2</sub> (about 1:200). Decomposes in moist air or water forming  $H_3PO_4$  and  $H_2S$ . Dissolves on heating in caustic soda to form a yellow sodium thiophosphate solution. Organic compounds containing oxygen are converted to sulfur-containing materials.

**REFERENCES:** 

A. Stock and B. Herscovici. Ber. dtsch. chem. Ges. <u>43</u>, 1223 (1910). A. Vos and E. H. Wiebenga. Acta Cryst. (Copenhagen) <u>8</u>, 217 (1955).

# Monothiophosphoric Acid H<sub>3</sub>PO<sub>3</sub>S

The thio derivatives of orthophosphoric acid—tetrathiophosphoric acid, trithiophosphoric acid, dithiophosphoric acid and monothiophosphoric acid—are unstable in aqueous media, forming  $H_2S$  and  $H_3PO_4$  by hydrolysis:

$$H_{3}PO_{x}S_{y} + n H_{2}O = H_{3}PO_{x+n}S_{y-n} + n H_{2}S$$

Of the four acids, however, monothiophosphoric acid is sufficiently stable that Klement obtained an 83% solution by reacting barium dithiophosphate with the stoichiometric quantity of  $H_2SO_4$ , followed by hydrolysis of the resulting dithiophosphoric acid to monothiophosphoric acid.

 $\begin{array}{rl} Ba_3(PO_2S_2)_2 + 3 \ H_2SO_4 &= 2 \ H_3PO_2S_2 + 3 \ BaSO_4 \\ (8 \ H_2O) \\ 810.4 & 294.2 & 260.3 & 700.3 \\ H_3PO_2S_2 + H_2O &= H_3PO_3S + H_2S \\ 130.1 & 18.0 & 114.1 & 34.1 \end{array}$ 

A 68.5-g. portion of barium dithiophosphate (see p. 571) is added to 50 ml. of  $H_2S$  water, saturated at 0°C; the exact quantity of 1 N  $H_2SO_4$  required by the analysis of the barium salt for precipitation of all the Ba (about 500 ml.), is rapidly dripped in, with ice cooling and vigorous stirring. The precipitated BaSO<sub>4</sub> is quickly centrifuged off and the solution is freed of  $H_2S$  in vacuum while cooling with ice. To convert the  $H_3PO_2S_2$  into  $H_3PO_3S$  hydrolytically, the solution is permitted to stand in ice for 12 hours. After this, the fresh quantity of  $H_2S$  produced by the hydrolysis is removed in vacuum. To concentrate the acid solution, which is about 3.5% at this point, the material is evaporated in high vacuum, using a ground-joint apparatus. The round-bottom flask containing the acid is constantly maintained at 0°C. The distilled water, which contains only traces of  $H_2S$ , is collected in a trap at -79°C. The residue is about 16 g. of oily liquid containing about 84%  $H_3PO_3S$ .

PROPERTIES:

If the concentrated acid is stored in a refrigerator at  $-2^{\circ}C$ , with exclusion of air, it remains clear. Introduction of air causes turbidity because of the precipitation of S due to oxidation of the H<sub>2</sub>S which results from the gradual hydrolysis of monothiophosphoric acid. Heating the concentrated acid to 50°C, results in vigorous liberation of H<sub>2</sub>S The acid is a vitreous solid at  $-60^{\circ}C$ . **REFERENCE:** 

#### R. Klement. Z. anorg, allg. Chem. 253, 242 (1947).

## Sodium Monothiophosphate

## $Na_{8}PO_{3}S\cdot 12\ H_{2}O$

Sodium monothiophosphate can be prepared in several ways. The oldest procedure is that of Wurtz (I) and is based on the hydrolysis of  $PSCl_3$  with NaOH. This procedure gives a very good yield.

A different method, reported by Kubierschky (II), uses  $P_2S_5$ , which is converted to a mixture of various sodium thiophosphates of sodium hydroxide. After isolation, the sodium thiophosphates are converted, by heating to 90°C in aqueous solution, into sodium monothiophosphate, which is then precipitated (see the first equation under monothiophosphoric acid, page 568). A very elegant method which gives an excellent yield was reported by Zintl and Bertram (III). This procedure is based on the addition of anhydrous sodium sulfide to anhydrous sodium trimetaphosphate (see page 552) at temperatures ranging from 500 to 750°C.

#### I. WURTZ PROCEDURE

 $\begin{array}{rrr} PSCl_{3} \,+\, 6\,NaOH \,=\, Na_{3}PO_{s}S \,+\, 3\,NaCl \,+\, 3\,H_{2}O \\ 169.4 & 240.0 & 180.0 & 175.4 & 54.0 \end{array}$ 

A quantity of  $PSCl_3(70 \text{ g.})$  is refluxed with a solution of 120 g. of NaOH in 400 ml. of water until all the  $PSCl_3$  has dissolved. After partial cooling, 100 ml. of alcohol is added and the mixture is thoroughly cooled in ice. The crude salt (approximately 125 g.) is dissolved in 90 ml. of water at a temperature not exceeding  $50^{\circ}C$  and, after filtration, the solution is gradually cooled, to ice temperature. The yield is 100 g. of analytically pure salt.

#### II. KUBIERSCHKY PROCEDURE

A solution of 100 g. of NaOH in 500 ml. of water is mixed with small portions of 100 g. of technical  $P_2S_5$ . The temperature of the mixture must not exceed 50°C. After filtration, 200 ml. of alcohol is added to the yellow solution, which is then cooled in ice. After several hours the precipitated salt is suction-filtered, washed (first with 50%, then with pure alcohol) and air-dried. The yield is approximately 220 g. of crude salt. This is then dissolved in 700 ml. of water. The solution is heated to approximately 70°C until the onset of vigorous evolution of  $H_2S$ . The solution is then immediately cooled to  $60^{\circ}$ C and kept at that temperature for 10 minutes to liberate the H<sub>2</sub>S. Then, 200 ml. of alcohol is added and the solution is cooled as rapidly as possible to ice temperature. The crude Na<sub>3</sub>PO<sub>3</sub>S is recrystallized as in method (I). It is not easy to recognize the end point of the hydrolysis of the various initial thiophosphates to the monothiophosphate. In addition it is desired to prevent the process from continuing until orthophosphate is formed. There, a single recrystallized several times to obtain the analytically pure salt. For this reason, the yield varies.

III. ZINTL AND BERTRAM PROCEDURE

 $\begin{array}{rl} Na_{3}P_{3}O_{9} \,+\, 3\,Na_{2}S \,=\, 3\,Na_{3}PO_{3}S \\ 305.9 & 234.2 & 540.1 \end{array}$ 

A mixture of  $Na_3P_3O_9$  and carefully dehydrated  $Na_2S$ , in the same ratio as in the above equation, is fused at 500 to 750°C. The mixture is placed in a carbon crucible and the fusion proceeds either in a stream of  $N_2$  or  $H_2$ , or in vacuum. The gray to yellow product contains up to 85% of anhydrous  $Na_3PO_3S$ . Solution of the melt in water and crystallization produces the dodecahydrate.

PROPERTIES:

Thin, hexagonal flakes that readily effloresce in air. Very readily soluble in hot water; aqueous solutions are strongly alkaline. M.p.  $60^{\circ}C$ .

**REFERENCES:** 

I. M. A. Wurtz. Ann. Chim. Phys. [3] 20, 473 (1847).

II. C. Kubierschky. J. prakt. Chem. 2 31, 93 (1885).

III. E. Zintl and A. Bertram. Z. anorg. allg. Chem. 245, 16 (1940).

# Sodium Dithiophosphate

 $Na_3PO_2S_2\cdot 11\ H_2O$ 

According to Kubierschky, sodium dithiophosphate, is obtained in the same manner as sodium monothiophosphate, i.e., through reaction of  $P_2S_5$  with sodium hydroxide and subsequent hydrolysis of the simultaneously formed higher thiophosphates (see page 569).

A mixture of various thiophosphates is prepared in the manner described on page 569 under (I). The crude salt is dissolved in

570

water. The solution is heated to  $50-55^{\circ}C$  and kept at that temperature until a sample of the solution produces a green but not a yellow color on mixing with a  $CoSO_4$  solution. However, it should not be heated to the point where a precipitate of monothiophosphate is formed with  $CaCl_2$ . The  $Na_2PO_2S_2$  is then precipitated with alcohol and is recrystallized from water at a temperature not exceeding  $40^{\circ}C$ .

PROPERTIES:

Formula weight 394.26. Hexagonal prismatic crystals are formed on slow cooling of the aqueous solution; rapid cooling yields needles up to 2 cm. long. M.p. 45-46 °C.

**REFERENCES:** 

C. Kubierschky. J. prakt. Chem. [2] <u>31</u>, 101 (1885).

E. Steeger and U. Seener. Z. anorg allg. Chem. 303, 21 (1960).

# **Barium Dithiophosphate**

 $Ba_3(PO_2S_2)_2 \cdot 8 H_2O$ 

Barium dithiophosphate, required in the preparation of free monothiophosphoric acid, is produced in accordance with the Kubierschky method by precipitation from  $Na_2PO_2S_2$  solution with  $BaCl_2$ :

 $\begin{array}{ccc} 2 \text{ Na}_{3}\text{PO}_{2}\text{S}_{2} + 3 \text{ BaCl}_{2} = \text{Ba}_{3}(\text{PO}_{2}\text{S}_{2})_{2} + 6 \text{ NaCl} \\ (11 \text{ H}_{2}\text{O}) & (2 \text{ H}_{2}\text{O}) \\ 788.5 & 732.9 & 810.4 & 350.7 \end{array}$ 

Stoichiometric quantities of the salts are separately dissolved in water, and the BaCl<sub>2</sub> solution is added to the other. The precipitated Ba<sub>3</sub>(PO<sub>2</sub>S<sub>2</sub>)<sub>2</sub> is suction-filtered, washed with alcohol and ether or acetone, and air-dried.

**REFERENCE:** 

C. Kubierschky. J. prakt. Chem. [2] 31, 103 (1885).

## Sodium Trithiophosphate

## Na<sub>3</sub>POS<sub>3</sub> · 11 H<sub>2</sub>O

Pure sodium trithiophosphate can be obtained by repeated fractional crystallization of the mixture of thiophosphates obtained from  $P_2S_5$  and NaHS by the Kubierschky method A solution of 100 g. of NaOH in 600 ml. of water is saturated with H<sub>2</sub>S. Then 60 g. of technical grade  $P_2S_5$  is gradually introduced at a temperature not exceeding 20°C. The solution is heated at 25°C for 10-15 minutes. It is then rapidly filtered and quickly cooled in ice. The salt, precipitated by addition of 200 ml. of alcohol or acetone, is suction filtered, washed successively with a small quantity of ice water, 25% alcohol, 50% alcohol, absolute alcohol and ether, and then air-dried. The yield is approximately 130 g.

The crude salt, with an average S content of 23%, is a mixture of varying quantities of di-, tr1-, and tetrathiophosphates. The salt is dissolved at 20°C in three times its weight of 10% Na<sub>2</sub>S solution, and the mixture is cooled in ice. Crystallization begins at approximately 4°C, but must sometimes be initiated with a few seed crystals of pure sodium dithiophosphate. After about one half hour, the mixture is suction filtered and washed as above (this is fraction 1, about 20 g.). Then 25 ml. of alcohol is added to the mother liquor and the precipitated crystals are suctionfiltered after one hour (this is fraction 2, about 50 g.). Fractions crystallized on continued addition of alcohol contain significant quantities of Na<sub>3</sub>PS<sub>4</sub> and are therefore discarded. They can sometimes be processed to Na<sub>3</sub>PO<sub>3</sub>S or Na<sub>3</sub>PO<sub>2</sub>S<sub>2</sub> (see page 569 f.)

Fractions 1 and 2 are combined and crystallized as above. The second fraction thus obtained is generally of such purity that simple recrystallization is sufficient to obtain analytically pure  $Na_3POS_3$ . Should the second fraction (obtained during the second fractional crystallization) turn out to be insufficiently pure, it must be further fractionated. The yield of pure salt is about 30-40 g.

## PROPERTIES:

Formula weight 410.32. Crystallizes well; colorless. The least stable sodium thiophosphate. Decomposes slowly when stored at  $-2^{\circ}$ C. After one year, one fifth of the salt is converted into Na<sub>2</sub>PO<sub>2</sub>S<sub>2</sub>.

#### REFERENCES:

C. Kubierschky. J. prakt. Chem. [2] <u>31</u>, 105 (1885). R. Klement. Z. anorg. Chem. <u>253</u>, 254 (1947).

## Sodium Tetrathiophosphate

## $Na_3PS_4 \cdot 8 H_2O$

According to Glatzel, sodium tetrathiophosphate is produced by treatment of  $P_2S_5$  with Na<sub>2</sub>S:

 $\begin{array}{c} P_2S_5 + 3 \text{ Na}_2S = 2 \text{ Na}_3PS_4 \\ (9 \text{ H}_2\text{O}) & (8 \text{ H}_2\text{O}) \\ 222.3 & 720.6 & 744.7 \end{array}$ 

Crystalline  $Na_2S \cdot 9H_2O$  (800 g.) is melted in a large porcelain dish and 80 g. of  $P_2S_5$  is then stirred in. In 10-20 minutes after solution has taken place, 800 ml. of hot water is added and the mixture is rapidly filtered through a large filter. The crude salt that precipitates after the mixture has been left standing for 24 hours is suction-filtered and recrystallized by the Klement method, which proceeds as follows. The salt is dissolved at a temperature not exceeding 10°C in a fivefold quantity of 2% Na<sub>2</sub>S solution in the precipitated by addition of an equal volume of alcohol, while cooling in ice. This operation must be rapid to prevent decomposition. The salt is washed with 50% alcohol, absolute alcohol and ether, and is then air-dried.

PROPERTIES:

Formula weight 372.34. Monoclinic crystals. Stable when stored in a closed container. Quickly hydrolyzes in aqueous solution (see equation under  $H_3PO_3S$  on p. 569).

REFERENCES:

E. Glatzel. Z. anorg. allg. Chem. <u>44</u>, 65 (1905). R. Klement. Z. anorg. allg. Chem. <u>253</u>, 246 (1947).

## **Tetraphosphorus Triselenide**

 $P_4Se_3$ 

According to Mai, tetraphosphorus triselenide is synthesized from the elements in Tetralin solution at  $200^{\circ}$ C:

 $4 P + 3 Se = P_4 Se_3$ 123.9 236.9 360.8

Ground (but not "powdery") fused Se (10 g.) is placed in a 100-ml. test tube. Then 50 ml. of Tetralin is added, together with 5 g. of white P wire which has been quickly dried with filter paper. The lower end of the test tube is immersed in a paraffin bath  $(215-220^{\circ}C)$ . As soon as the P melts, the tube contents are mixed with an L-shaped rod until the mass acquires a uniform consistency. The tube is then lowered into the bath so that all the Tetralin is surrounded by the paraffin and begins to boil. At a

point at which frequent stirring no longer produces P vapor, the mixture is filtered hot through glass wool (the funnel in which the glass wool is placed has a very short stem). The product is collected in a well-heated Erlenmeyer flask. The crude product is extracted by boiling twice with 50 ml. of Tetralin. After several hours, long, prismatic crystals separate out of the first filtrate and the extracts. The Tetralin is decanted, since it can be reused. The crude product is washed with alcohol until no further emulsion is produced. (After settling, the alcohol can be decanted and reused.) The crude product is then placed in a filter with fresh alcohol, washed with ether and chloroform, and carefully suctionfiltered without, however, permitting it to dry out. It is then extracted in a Soxhlet apparatus with CS2. The P2Se3 deposits on the tube in a firm crystalline crust. The mother liquor is de-canted and the crystals are washed with CCl<sub>4</sub>. A small additional quantity of P<sub>4</sub>Se<sub>3</sub> can be obtained from the mother liquor by precipitation with ether. Recrystallization from benzene yields long, transparent, almost odorless needles. The product must be stored in a desiccator. The yield is approximately one third of theoretical.

**PROPERTIES:** 

Long, yellow to red-orange needles which evolve  $H_2Se$  in air. Soluble in cold  $CS_2$ ; soluble in warm chloroform, carbon tetrachloride, acetone, benzene and toluene. M.p. 242°C, b.p. 360-400°C.

REFERENCE

J. Mai. Ber. dtsch. chem. Ges. 61, 1807 (1928).

# Triphosphorus Pentanitride

## $P_3N_5$

According to Stock and Hoffman, triphosphorus pentanitride is prepared by heating  $P_2S_5 \cdot 7NH_3$  to bright red heat in a stream of NH<sub>3</sub> or H<sub>2</sub>. According to Moureu and De Ficquelmont, the compound can also be formed by heating  $(PNCl_2)_n$ (see p. 575) at 825°C in a stream of NH<sub>3</sub>. Both reactions involve several intermediate steps.

I. The addition compound  $P_2S_5 \cdot 7NH_3$ , formed at  $-20^{\circ}C$  by treating  $P_2S_5$  (see p. 567) with gaseous, dry NH<sub>3</sub>, is gradually heated in a stream of NH<sub>3</sub> to 230°C, followed by careful heating in a stream of H<sub>2</sub> or N<sub>2</sub> to a higher temperature. First,  $(NH_4)_2S$  is liberated, then P and S are evolved, and after heating to bright red heat only  $P_3N_5$  remains. To remove the last traces of S, the heat must be

increased steadily almost to the temperature of decomposition of the  $P_3N_5$ . It is best to use a Pyrex tube, open at both ends, which fits tightly inside an outer Pyrex tube. This is preferred to a porcelain boat since the substance is more exposed to the gas than in the boat. Hydrogen is preferred to nitrogen, since it reduces the extensive decomposition of the product. If the temperature is too high, an odor of  $NH_3$  will be noticeable, an indication that the temperature must be immediately lowered.

II. Some  $(PNCl_2)_3$  is heated in a stream of NH<sub>3</sub>. The intermediate chloride-containing phospham evolves HCl in vacuum at 170-180°C and is then heated to 600°C. When the temperature is further increased to 800-825°C, in a stream of NH<sub>3</sub>, pure P<sub>3</sub>N<sub>5</sub> is obtained.

PROPERTIES:

Formula weight 162.96. White, odorless, tasteless; decomposes into its elements in vacuum at high temperature. Insoluble in all solvents. Heating with water in a sealed tube at  $180^{\circ}$ C decomposes  $P_3N_5$ , forming  $H_3PO_4$  and  $NH_3$ . Oxygen affects it (ignition) only at temperatures above  $600^{\circ}$ C. d ( $18^{\circ}$ C) 2.51.

**REFERENCES:** 

- I. A. Stock and B. Hoffmann. Ber. dtsch. chem. Ges. <u>36</u>, 317 (1903).
- II. H. Moureu and A. M. de Ficquelmont. Compt. Rend. Hebd. Séances Acad. Sci. 198, 1417 (1934).

## Phosphonitrilic Chlorides

(PNCl<sub>2</sub>)<sub>n</sub>

Phosphonitrilic chlorides of varying degrees of polymerization (n = 3-7) are formed through reaction of  $PCl_{5}$  with NH<sub>4</sub>Cl at elevated temperatures:

$$\begin{array}{l} PCl_{5} + NH_{4}Cl = PNCl_{2} + 4 HCl \\ 208.3 \quad 53.5 \quad 115.9 \quad 145.9 \end{array}$$

Schenck and Roemer performed the reaction at  $135^{\circ}$ C, using tetrachloroethane as the solvent for the PCl<sub>5</sub>. Upon recrystallization from petroleum ether, the crude product is vacuum-distilled, yielding the trimer or tetramer. Audrieth et al. and Steinmann et al. report heating a mixture of the reactants directly to temperatures up to  $160^{\circ}$ C, and they describe various separation methods for the different polymers.

I. A mixture of 400 g. of PCl<sub>5</sub> and 120-130 g. of finely powdered, dry NH<sub>4</sub>Cl is heated to 135°C in one liter of tetrachloroethane. The round-bottom flask containing the mixture is placed in an oil bath. The flask is equipped with a reflux condenser, the latter having a CaCl<sub>2</sub> tube and a delivery tube. The solution is boiled gently and gradually evolves HCl. Although the main reaction is complete in seven hours, heating is continued for another 13 hours. After cooling, the unreacted NH<sub>4</sub>Cl is filtered off and the solvent is distilled at about 11 mm. and 50°C. The initially liquid residue solidifies to a crystalline mass containing traces of an oil and weighing approximately 220 g. This mass is placed in a roundbottom 500-ml. flask equipped with a reflux condenser. It is extracted by boiling 10 minutes with 200-250 ml. of petroleum ether (b.p. 40-60°C). This is repeated five or six times. The clear petroleum ether layer is decanted from the oily residue and transferred to a distillation apparatus. The petroleum ether is then distilled off on a water bath. The petroleum ether can be reused for subsequent extractions. To determine when further extraction becomes unnecessary, a glass rod is dipped into the extract, the solvent is allowed to evaporate, and the thickness of the layer of  $(PNCl_2)_n$  (n = 3, 4) is observed. Partial evaporation of the petroleum ether solution of (PNCl<sub>a</sub>)<sub>n</sub> on a water bath under atmospheric pressure, followed by cooling of the concentrate, produces crystals of the trimer and tetramer. The crystals are suctionfiltered, washed with a small quantity of petroleum ether, and separated from adhering liquid by pressing. After drying vith filter paper, the crystals are placed in a distillation flask and distilled in vacuum with an oil bath. The flask is immersed in the bath as far as possible. At 10 mm., two fractions are obtained. These boil at 124°C and 185°C. The distillates are separately recrystallized from benzene or chloroform. Approximately 50 g. of the trimer and 15 g. of the tetramer are obtained.

II. Instead of starting with  $PCl_5$ , Schenck and Roemer recommend that an equivalent quantity of  $PCl_3$  be dissolved in tetrachloroethane and the necessary quantity of  $Cl_2$  added. The reaction with  $NH_4Cl$  is then carried out.

III. An intimate mixture of 52.1 g. of  $PCl_5$  and 50-100 g. of  $NH_4Cl$  is prepared, covered with a 2- to 7-cm. protective layer of  $NH_4Cl$ , and heated 4-6 hours in an oil bath at 145-160°C. Most of the  $PNCl_2$  trimer sublimes and deposits in the cooler section of the apparatus. The remaining trimer and tetramer are quantitatively extracted with petroleum ether (b.p.  $50-70^{\circ}C$ ). The higher homologs are obtained as viscous oils or rubbery materials by treatment of the residue with benzene, chloroform or carbon tetrachloride.

The trimer is distilled from the residue of the petroleum ether solution at 12-14 mm. and 140°C. It is recrystallized from acetic acid. For further purification, it is sublimed at 1 mm. and  $100 \pm 5$ °C.

IV. The polymers can also be separated in the following manner. 1) The higher homologs are considerably more soluble in cold benzene than the trimer or the tetramer. 2) The lower homologs are more soluble in anhydrous glacial acetic acid than the higher ones. 3) Only the trimer is carried over in a steam distillation, as the other homologs are hydrolyzed. 4) The trimer and the tetramer can be separated by distillation at reduced pressure (see method I).

PROPERTIES:

The phosphonitrilic chlorides polymerize at temperatures above  $255^{\circ}$ C. The polymerization is complete after six hours of heating. The infusible, elastic mass formed is completely colorless and transparent. It swells in benzene, forming a colloid. It resists cold acids and alkali, is gradually attacked by boiling water, and, because of substitution of Cl by OH it transforms into cyclic acids. On long exposure to air, the material loses its elasticity and decomposes to a friable mass. The polymers consist of long chain molecules, stretched, frozen samples of which have a fiber pattern similar to that of rubber.

The phosphonitrilic chlorides have an aromatic odor and damage the respiratory organs. The decomposition products cause eye and respiratory difficulties and cause apathy after prolonged exposure. Antidote: Inhalation of  $NH_3$ -containing air.

Audrieth, Steinmann, and Toy give the following data for the homologous phosphonitrilic chlorides:

	Density	M.p., °C	B.p., °C	Crystal system
Trimer	1.98	114	127 (13 mm.) 256 (760 mm.)	rhombic Lines in which P and N atoms
Tetramer	2.18	123.5	188 (13 mm.) 328.5 (760 mm.)	tetragonal ) alternate
Pentamer Hexamer Heptamer		40.5 - 41 90 - 91 18	223 - 224.3 (13 mm.) 261 - 263 (13 mm.) 289 - 294 (13 mm.)	rhombic, chains

According to the same authors, the solubilities in g./100 g. of solvent are as follows:

	Ether	Dioxane	Benzene	Toluene	Xylene	Alkanes	CC14	CS <sub>2</sub>
Trimer	46.37	29.55	55.01	47.3	38.85	27.9	38.88	52.05
Tetramer	12.4	8.23	21.42	17.8	13.85	8.39	16.55	22

**REFERENCES:** 

- R. Schenck and G. Römer. Ber. dtsch. chem. Ges. 57, 1345 (1924).
- R. Steinmann, F. B. Schirmer and L. F. Audrieth. J. Amer. Chem. Soc. 64, 2377 (1942).
- L. F. Audrieth, R. Steinmann and A. D. F. Toy. Chem. Rev. <u>32</u>, 109 (1943).

#### **Phosphonitrilic Bromides**

## (PNBr<sub>2</sub>)n

Phosphonitrilic bromides (n = 3, 4), discovered by Besson, are obtained, according to Grimme, from a solution of PBr<sub>5</sub> in tetrachloroethene by reaction with NH<sub>4</sub>Br:

 $PBr_5 + NH_4Br = PNBr_2 + 4 HBr$ 430.6 98.0 204.8 323.7

Bode has described the preparation and separation of the homologs in great detail.

A sample of  $PBr_3$  is dissolved in tetrachloroethene and the quantity of  $Br_2$  necessary for its conversion to  $PBr_5$  is added. Following this reaction the stoichiometric quantity of  $NH_4Br$  is added. The mixture is heated to the boiling point of the solvent (147°C). At this temperature,  $PBr_5$  decomposes partially, liberating  $Br_2$ , which therefore must be replenished from time to time. After the reaction is complete, the solvent is evaporated and the remaining crystalline slurry is fractionally crystallized from benzene. The solubility of the tetramer (a reaction byproduct) differs only slightly from that of the trimer. A simple separation method consists in slowly evaporating the benzene solution and then separating the characteristically different crystals by manual sorting.

PROPERTIES:

The trimer PNBr<sub>2</sub> forms tabular, well-formed lamellae of the rhombic crystal system. M.p. 191°C. The tetramer forms prismatic crystals which belong to the tetragonal system. M.p. 202°C.

REFERENCES:

A. Besson. Compt. Rend. Hebd. Séances Acad. Sci. <u>114</u>, 1479 (1892).
W. Grimme. Thesis, Münster, 1926 (cited by H. Bode).
H. Bode. Z. anorg. allg. Chem. 252, 113 (1943).

# Monoamidophosphoric Acid H2PO3NH2

Monoamidophosphoric acid is prepared by the Stokes method. Phenol and  $POCl_3$  are reacted to form the diphenyl ester of monochlorophosphoric acid. The latter is reacted with alcoholic ammonia to give the phenyl ester of monoamidophosphoric acid. Monoamidophosphoric acid is formed on saponification of the ester with KOH. It is isolated as the potassium salt, from which, according to Klement and Becht, free amidophosphoric acid can be obtained by reaction with  $HClO_4$ :

$POCl_3 + 2$	$C_6H_5OH =$	POCl(OC <sub>6</sub> H <sub>5</sub> ) <sub>2</sub>	+ 2 HCl
153.3	188.2	268.6	72.9
POCl(OC <sub>6</sub> H <sub>5</sub> );	$2 + 2 NH_3 =$	PONH <sub>2</sub> (OC <sub>6</sub> H <sub>2</sub>	$_{5})_{2}$ + NH <sub>4</sub> Cl
268.6	34.1	249.2	53.5
PONH <sub>2</sub> (OC <sub>6</sub> H	5)2 + 2 KOH	$= K_2 PO_3 NH_2$	$+ 2 C_6 H_5 OH$
249.2	112.2	173.2	188.2

$K_2PO_3NH_2 +$	CH <sub>3</sub> COOH =	= KHPO <sub>3</sub> NH <sub>2</sub>	$+ CH_{3}COOK$
173.2	60.1	135.1	98.1
KHPO <sub>3</sub> N	$H_2 + HClO_4$	= H <sub>2</sub> PO <sub>3</sub> NH <sub>2</sub>	+ KClO4
135.1	100.5	97.0	138.6

According to Klement, free amidophosphoric acid can also be obtained by reaction of a solution of the acid potassium salt with a cation exchanger that is not too strongly acidic.

A quantity of freshly distilled  $POCl_3$  (154 g.) is refluxed for 2-3 hours with 188 g. of freshly distilled phenol. The product is kept overnight in vacuum over NaOH to remove additional HCl. The oily liquid is added by drops to a minimum of 300 ml. of absolute alcohol, presaturated with NH<sub>3</sub> at 0°C. The reaction flask is ice-cooled and vigorously stirred. The mixture must be ammoniacal at the end of the reaction. The alcoholic solution is poured into five times its volume of ice water. The precipitate of the diphenyl ester of monoamidophosphoric acid is filtered off by suction, freed of adhering oil by pressing on clay, and dried in vacuum. The crude ester (approximately 160 g.) is recrystallized from 200 ml. of alcohol. The yield is approximately 100 g. (m.p. 148°C). An additional 50 g. of impure ester can be obtained by precipitation of the alcoholic mother liquor with water.

Small portions of 125 g. of the ester are gradually added to a hot solution of 140 g. of KOH in 280 ml. of water. The ester dissolves with a vigorous reaction. The saponification is terminated after boiling for a maximum of five minutes. The mixture is carefully cooled in ice and then acidified with 50% acetic acid. After standing in ice for one hour, the white, crystalline precipitate of KHPO<sub>3</sub>NH<sub>2</sub> is suction-filtered and is successively washed with 50% alcohol, absolute alcohol and ether. A small amount of the potassium salt can be recovered from the aqueous phase of the filtrate upon addition of alcohol. The yield is 50 g.

A solution of 13.5 g. of the potassium salt in 125 ml. of ice water is filtered, if necessary, and 100 ml. of 10% HClO<sub>4</sub> is added dropwise, while the flask is cooled in an ice bath. The KClO<sub>4</sub> precipitate is suction-filtered after a short time, and 700 ml. of alcohol is added to the filtrate. After long standing in an ice bath, the H<sub>2</sub>PO<sub>3</sub>NH<sub>2</sub>, somewhat contaminated with KClO<sub>4</sub>, is suction-filtered, washed with alcohol and ether, and air dried. The yield is approximately 8 g. The acid is purified by dissolving in 150 ml. water; the filtered solution is precipitated with an equal volume of alcohol, refiltered after standing in an ice bath, and washed as above. The yield of pure acid is 4 g.

To liberate the acid with a cation exchanger, the latter is prepared in accordance with the instructions given on p. 555, and the procedure described there is followed. A solution of 10 g, of  $KHPO_3NH_2$  in 100 ml. of water is added to the resin. After 10 minutes it is decanted and the resin is rinsed with 50-ml. and 25-ml. portions of pure water. The filtered and combined solutions are then precipitated with 600 ml. of alcohol. The yield of free  $H_2PO_3NH_2$  is 5 g. (about 60% of theoretical) of comparatively pure material. It can be further purified by reprecipitation as above.

Other Preparative Method: The Adams procedure may be used for hydrogenation of  $PO(NH_2)(OC_6H_5)_2$  with platinum oxide; this easily yields H<sub>2</sub>PO<sub>3</sub>NH<sub>2</sub> (M. Becke-Goehring and J. Sambeth):  $PO(NH_2)(OC_6H_5)_2 + 16 H = H_2PO_3NH_2 + 2C_6H_{12}$ . In the presence of 200 mg. of platinum oxide, five grams of PO(NH<sub>2</sub>)(OC<sub>6</sub>H<sub>5</sub>)<sub>2</sub>, dissolved in 100 ml. of anhydrous methanol, absorb approximately 95% of the calculated quantity of hydrogen in 10 to 15 hours. Toward the end of the reaction, another 200 g. of platinum oxide is added. The free acid, along with the platinum, is carefully filtered by suction through a compact filter. The acid is washed with methanol without coming in contact with the air. The acid is dissolved on the filter with as little water as possible and is then precipitated with ethanol. After standing in an ice bath for one hour, the acid is suction-filtered and washed with acetone and ether. The yield is 1 g (51%). This procedure may also be employed to produce other substituted phosphoric acids.

#### PROPERTIES:

Exists only in the anhydrous form. Colorless, prismatic crystals, easily soluble in water. The aqueous solution hydrolyzes, forming

 $\rm NH_4H_2PO_4$ . It is therefore impossible to obtain completely pure  $\rm H_2PO_3NH_2$  (Klement and Hille). Rapidly hydrolyzes on boiling the solution in the presence of dilute sulfuric acid. Stable in sealed ampoules. Produces a white, crystalline precipitate with AgNO<sub>3</sub>. Heating of the free acid to  $110^{\circ}C$  yields an ammonium polyphosphate (Goehring and Sambeth).

**REFERENCES:** 

- N. H. Stokes. Amer. Chem. J. 15, 198 (1893).
- R. Klement and K.-H.Becht. Z. anorg. allg. Chem. 254, 217 (1947).
- R. Klement. Z. anorg. allg. Chem. 260, 267 (1949).
- R. Klement and V. Hille. Z. anorg. allg. Chem. 289, 89 (1957). M. Goehring and J. Sambeth. Chem. Ber. 90, 232 (1957).
- M. Becke-Goehring and J. Sambeth. Chem. Ber. <u>90</u>, 2075 (1957; Angew. Chem. 70, 594 (1958).

## Disodium Monoamidophosphate

#### $Na_2PO_3NH_2 \cdot 6 H_2O$

Disodium monoamidophosphate is formed on saponification of the diphenyl ester of monoamidophosphoric acid (see p. 579) with the stoichiometric quantity of NaOH (Klement and Biberacher):

$PO(NH_2) (OC_6H_5)_2 +$	4 NaOH	$= Na_2PO_3HN_2$	$+ 2 C_6 H_5 ONa$	$+ 2 H_2O$
		$(6 H_2 O)$		
249.2	160.0	249.1	232.2	36.0

A 25-g. quantity of the diphenyl ester of monoamidophosphoric acid is boiled in a solution of 16 g. of NaOH in 60 ml. of water for 10 minutes. Crystallization of the salt commences on cooling the clear solution to 0°C. This is aided by addition of ice-cold ethanol (200 ml. is needed for complete precipitation). The Na<sub>2</sub>PO<sub>3</sub>NH<sub>2</sub> •  $6H_2O$  obtained is already quite pure. To reprecipitate, or to obtain larger crystals, it is dissolved in a small quantity of water together with pellets of NaOH. Then ethanol is added to the filtered solution at room temperature. While the salt initially precipitates as an oily substance, larger crystals are formed after brief stirring. The crystals are washed with ethanol and ether and dried in vacuum. The yield is 20 g.

PROPERTIES:

The salt effloresces in air and is simultaneously decomposed by  $CO_2$ . Therefore it must be stored in airtight containers. Readily soluble in water.

**REFERENCE:** 

R. Klement and G. Biberacher. Z. anorg. allg. Chem. 283, 246 (1956).

## Diamidophosphoric Acid

#### $HPO_2(NH_2)_2$

In the Stokes method (I), the synthesis of diamidophosphoric acid starts from the phenyl ester of dichlorophosphoric acid obtained from POCl<sub>3</sub> and phenol. This is converted with aqueous NH<sub>3</sub> to the phenyl ester of diamidophosphoric acid. This ester is then saponified with Ba (OH)<sub>2</sub>; the Ba<sup>2+</sup> is precipitated as BaCO<sub>3</sub>, and the silver salt of diamidophosphoric acid is then obtained by precipitation with AgNO<sub>3</sub>. After reprecipitation of the silver salt, treatment with HBr results in very pure, free diamidophosphoric acid. The phenyl ester of diamidophosphoric acid can also be saponified with potassium hydroxide, and after acidification with acetic acid, the free diamidophosphoric acid can be precipitated with ethanol, yielding a somewhat less pure product (method II).

 $POCl_3 + C_6H_5OH = POCl_2(OC_6H_5) + HCl$ 94.1 211.0 153.3 36.5  $POCl_2(OC_6H_5) + 4 NH_3 = PO(NH_2)_2(OC_6H_5) + 2 NH_4Cl$ 211.0 68.1 172.1107.0  $2 PO(NH_2)_2(OC_6H_5) + Ba(OH)_2 = Ba[PO_2(NH_2)_2]_2 + 2 C_6H_5OH$ (8 H<sub>2</sub>O) 344.2 327.4 188.2315.5  $AgPO_2(NH_2)_2 + HBr = HPO_2(NH_2)_2 + AgBr$ 202.980.9 96.0 187.8  $PO(NH_2)_2(OC_6H_5) + KOH = KPO_2(NH_2)_2 + C_6H_5OH$ 172.156.1134.1 94.1  $KPO_2(NH_2)_2 + CH_3COOH = HPO_2(NH_2)_2 + CH_3COOK$ 96.0 134.1 60.1 98.1

A mixture of 95 g. of freshly distilled phenol and 160 g. of freshly distilled POCl<sub>3</sub> is refluxed until HCl evolution ceases (about eight hours). The product is distilled at 14 mm. and the fraction boiling at  $115-120^{\circ}$ C is collected. The oily liquid (approximately 180 g.) is added dropwise to 750 ml. of ice-cold, vigorously stirred concentrated ammonia. The precipitated phenyl ester of diamidophosphoric acid is rapidly filtered off, washed with ice water and dried on clay plates. A quantity (70-80 g.) of the crude ester is recrystallized twice from 1000-1250 ml. of 95% ethanol. The m.p. of the pure ester is 188°C.

I. In this method not more than 30 g, of the phenyl ester of diamidophosphoric acid should be used per run since otherwise the yield of silver salt will diminish. One part of ester is boiled for 10 minutes with two parts of Ba  $(OH)_2 \cdot 8 H_2O$  and ten parts of water. After cooling, CO<sub>2</sub> is bubbled through until a neutral reaction is obtained, and the BaCO<sub>3</sub> precipitate is immediately filtered off. The filtrate is treated with a 10% AgNO<sub>3</sub> solution until no further precipitate is formed. The supernatant is decanted through a filter, the silver residue is suspended in 150-200 ml. of water, and ammonia is added until the solution acquires a definite ammonia odor. Disregarding the incomplete solution, nitric acid (1:1) is added dropwise without intermediate filtration until a pH of 7. At this point, the solution is filtered through a fritted glass filter and the precipitate is discarded. Dropwise addition of nitric acid to the filtrate is continued until complete precipitation of pure primary silver diamidophosphate. The latter appears curdy at first, but soon assumes a crystalline form. No excess nitric acid should be used, because part of the silver salt may dissolve. The salt is filtered by suction, washed with ethanol and dried in vacuum. The yield is 20 g.

A 10-g. sample of  $AgPO_2(NH_2)_2$  is suspended in 50 ml. of water. Hydrobromic acid (57%) is added in drops to the stirred solution until complete reaction is achieved (0.44 ml. of acid per gram of silver salt). The silver bromide is filtered off and 150 ml. of ethanol is gradually added to the filtrate. Crystalline HPO<sub>3</sub> (NH<sub>2</sub>)<sub>2</sub> precipitates after brief ice cooling. The crystals are suctionfiltered, washed with 70% ethanol and dried in vacuum. The yield is 3.5 g.

II. A mixture of 30 g. of the phenyl ester of diamidophosphoric acid and a hot solution of 30 g. of KOH in 30 ml. of water is kept boiling for five minutes. It is then cooled in ice, and a mixture of 30 g. of acetic acid and 10 ml. of water is added, followed by 300 ml. of ethanol. The distinctly acid solution is allowed to stand in ice for a while, and the precipitate of diamidophosphoric acid is suction-filtered and washed with ethanol and ether. Upon drying in air, a yield of 11 g. is obtained (65%, based on the ester). To purify, the acid is dissolved in a small amount of water and reprecipitated with ethanol.

Alternate Method: When 5.1 g. of  $PO(NH_2)_3$  (see below) is heated on a water bath with 100 ml. of 10% sodium hydroxide for two hours, the following reaction takes place:  $PO(NH_2)_3 + NaOH =$  $NH_3 + NaPO_2(NH_2)_2$ . The cooled solution is mixed with 20% perchloric acid until a pH of 6 is reached and the silver salt is precipitated with 10% AgNO<sub>3</sub> solution; the salt is then treated as outlined in method I. [R. Klement and O. Koch, Ber. dtsch. chem. Ges. <u>87</u>, 333 (1954)]. **PROPERTIES:** 

Colorless, hexagonal stars, prisms or short needles. More stable than the monoamidophosphoric acid; however, is transformed into ammonium hydrogen monoamidophosphate if exposed to air for several months:  $HPO_2 (NH_2)_2 + H_2O = NH_4HPO_3NH_2$ . Thus, it must be stored in sealed glass ampoules. Liberates  $CO_2$  from carbonates in the cold. Over an eight-hour period,  $NH_4HPO_3NH_2$  is formed in the aqueous solution of the free acid. After 24 hours, this is transformed to  $(NH_4)_2HPO_4$ . M.p. ~ $100^{\circ}C$  (dec.).

**REFERENCES:** 

- I. R. Klement, G. Biberacher and V. Hille. Z. anorg. allg. Chem. 289, 80 (1957).
- II. N. H. Stokes. Amer. Chem. J. 16, 123 (1894).

# **Phosphoryl Triamide**

## PO(NH<sub>2</sub>)<sub>3</sub>

I. Phosphoryl triamide can be prepared via reaction of  $NH_3$  with a well-cooled chloroform solution of  $POCl_3$  (Wetroff):

 $\begin{array}{rl} \mathrm{POCl}_3 \,+\, 6\,\mathrm{NH}_3 \,=\, \mathrm{PO}(\mathrm{NH}_2)_3 \,+\, 3\,\mathrm{NH}_4\mathrm{Cl} \\ 153.3 & 102.2 & 95.1 & 160.5 \end{array}$ 

To separate the products, the  $NH_4Cl$  is reacted with diethylamine to yield  $NH_3$  and diethylammonium chloride, which is soluble in  $CHCl_3$ :

$$NH_4Cl + NH(C_2H_5)_2 = NH_3 + [(C_2H_5)_2NH_2]Cl.$$

The unreacted PO  $(NH_2)_3$ , which is insoluble in CHCl<sub>3</sub>, is recrystallized from methanol (Klement and Koch).

A three-neck, two-liter flask is provided with a mercury-seal stirrer, a cooled dropping funnel (see Fig. 198) and a ground glass cap with an inlet and a connecting tube. The latter is attached to a drying tube filled with solid NaOH. The flask is filled with 1.5 liters of freshly distilled and CaCl<sub>2</sub>-dried CHCl<sub>3</sub> and is cooled in an ice-salt mixture to  $-15^{\circ}$ C. Then a fast stream of well-dried NH<sub>3</sub> is bubbled through for three hours, until the CHCl<sub>3</sub> is fully saturated. The cooling jacket of the dropping funnel is then filled with ice-salt mixture and the funnel itself is filled with a solution of 60 g. (37 ml.) of freshly distilled POCl<sub>3</sub> in 100 ml. of CHCl<sub>3</sub>. The ammonia solution is stirred as vigorously as possible; it is chilled

to  $-15^{\circ}$ C and kept at that temperature with very efficient cooling, and the POCl<sub>3</sub> is added dropwise over a period of two hours, while continuing the passage of NH<sub>3</sub>. Slow bubbling of NH<sub>3</sub> is continued for another hour. The cold bath is then removed, the mixture is allowed to stand overnight, and the precipitate is rapidly suctionfiltered. It is washed with dry CHCl<sub>3</sub> and dried in vacuum. The yield is 100 g.

The product is mixed with 225 ml. of dry  $CHCl_3$  and 160 g. (230 ml.) of diethylamine and heated four hours at 60°C in a flask equipped with a reflux condenser. Higher temperatures should be avoided because decomposition will occur. Then 200 ml. of  $CHCl_3$  is added and the PO  $(NH_2)_3$  product is filtered off. It is washed with dry  $CHCl_3$  until both product and wash liquid no longer give a positive chloride test; the product is then dried in vacuum. The yield is 34 g.

To recrystallize the crude product, it is heated on a water bath with 150 ml. of absolute methanol in the complete absence of moisture. The solution is filtered hot and the pure PO  $(NH_2)_3$ , which precipitates on cooling, is suction-filtered. The mother liquor is discarded because it still contains some chloride. The undissolved crude product remaining is then treated in a similar manner with 200 ml. of absolute methanol. The mother liquor from this crystallization is reused for dissolving and recrystallizing the remainder of crude product, a procedure that requires repetition of the above steps up to three times. The combined crystals are dried in vacuum. The product is chromatographically pure. The yield is 26 g. (70%, based on POCl<sub>a</sub>).

II. According to Goehring and Niedenzu, PO (NH<sub>2</sub>)<sub>3</sub> can be obtained from POCl<sub>a</sub> via direct reaction with liquefied NH<sub>a</sub>. The apparatus shown in Fig. 199 is used. Tube a is half filled (by condensation) with 100-150 ml, of well-dried, pure NH<sub>3</sub>. Then  $h_1$  is closed, while  $h_2$  remains open. However, a drying tube filled with KOH is attached at  $h_a$  to protect a against atmospheric moisture. The temperature is reduced to  $-80^{\circ}$ C and the POClais slowly forced with well-dried compressed N<sub>2</sub> from the dropping funnel into q. Throughout this process, ha remains closed. After 20-40 g, of POCl, has been added, the dropping-funnel stopcock is closed, tube b is evacuated via  $h_{\rm s}$  to approximately 3 mm.,  $h_{\rm z}$  is closed and tube b is cooled to -80°C. The cooling bath is then removed from tube a, and  $h_3$  and  $h_{4}$  are opened. Within one minute, the liquid from a begins to flow through  $h_3$ , s and  $h_4$  into tube b. The frit f in tube a retains the already quite pure PO (NH<sub>2</sub>)<sub>3</sub>, while most of the NH<sub>4</sub>Cl remains dissolved in the NH<sub>3</sub> and is transferred into b. Stopcocks  $h_3$  and  $h_4$  are again closed and  $h_6$  is connected to  $h_2$ . The cooling bath is removed from tube b and transferred to tube a. Stopcocks  $h_{c}$  and  $h_2$  are again opened and the NH<sub>3</sub> is again condensed in a. Then  $h_2$ and  $h_{\rm f}$  are closed and the filtration is repeated. This procedure is



Fig. 198. Cooled dropping funnel for preparation of PO(NH<sub>2</sub>)<sub>3</sub>. The upper bulb has a capacity of 50 ml.; the inner tube has a diameter of 2 cm. The outer cooling jacket has a diameter of 7 cm. and is 15 cm. high.

Fig. 199. Preparation of phosphoric acid amides according to Goehring and Niedenzu. a) reaction tube; b) condensation trap and storage tube; f) glass frit;  $h_1-h_2$ ) stopcocks; s) ball and socket joint.

repeated five times. Finally,  $h_3$  is closed and dry N<sub>2</sub> is passed through  $h_1$  and  $h_2$  into a, which is then heated on a water bath to remove residual NH<sub>3</sub>. The pure PO(NH<sub>2</sub>)<sub>3</sub> can be removed from a by shaking. The yield is 94%.

**PROPERTIES:** 

Colorless crystals, probably belonging to the monoclinic system. Insoluble in ethanol, readily soluble in methanol, very readily in water. However, the aqueous solution is unstable because it converts to orthophosphate via amidophosphates. Heating with sodium hydroxide produces diamidophosphate (see p. 583). Introduction of HCl into a  $-15^{\circ}$ C suspension of PO(NH<sub>2</sub>)<sub>3</sub> in ether produces imidodiphosphoric acid tetramide (NH<sub>2</sub>)<sub>2</sub>P(O)-NH-P(O)(NH<sub>2</sub>)<sub>2</sub>; diimidotriphosphoric acid pentamide (NH<sub>2</sub>)<sub>2</sub>P(O)-NH-P(O)(NH<sub>2</sub>)-NH-P(O)(NH)<sub>2</sub> is formed at 30°C. Both substances are formed when PO(NH<sub>2</sub>)<sub>3</sub> is heated to 120°C in toluene. In moist air, PO(NH<sub>2</sub>)<sub>3</sub> converts within a few weeks to ammonium hydrogen monoamidophosphate. Thus, it must be stored in sealed glass vessels. With POCl<sub>3</sub>, forms a chlorine-containing intermediate which, after treatment with NH<sub>3</sub>, yields the amide-imide of orthophosphoric acid [H<sub>2</sub>N-P(O) NH]<sub>n</sub>.

**REFERENCES:** 

- I. G. Wetroff. Thesis, Paris, 1942; R. Klement and O. Koch. Chem. Ber. 87, 333 (1954).
- II. M. Goehring and K. Niedenzu. Chem. Ber. <u>89</u>, 1768, 1771, 1774 (1956).

# Thiophosphoryl Triamide

## PS(NH<sub>2</sub>)<sub>3</sub>

I. Thiophosphoryl triamide is obtained similarly to phosphoryl triamide (see p. 584) via reaction of  $NH_3$  and  $PSCl_3$  in chloroform solution. The reaction flask must be well cooled (Klement and Koch):

 $\begin{array}{rrrr} PSCl_3 \ + \ 6 \ NH_3 \ = \ PS(NH_2)_3 \ + \ 3 \ NH_4Cl \\ 169.4 \ 102.2 \ 111.1 \ 160.5 \end{array}$ 

The products are separated with diethylamine.

The apparatus of Fig. 198 is used. A solution of 60 g. of freshly distilled PSCl<sub>3</sub> in 100 ml. of CHCl<sub>3</sub> is added by drops to a solution of NH<sub>3</sub> in CHCl<sub>3</sub>; the addition time is two hours, and vigorous stirring and efficient cooling to maintain the temperature at  $-15^{\circ}$ C are imperative. Further procedure is the same as given in method I for phosphoryl triamide. The product (approximately 105 g.) is treated with 225 ml. of dry CHCl<sub>3</sub> and 100 g. of diethylamine. The yield of crude PS(NH<sub>2</sub>)<sub>3</sub> is approximately 32 g. Recrystallization from methanol is effected in the same way as for PO (NH<sub>2</sub>)<sub>3</sub>, but instead of 150 or 200 ml., 100-150 ml. of solvent is used due to the greater solubility here. The yield is 26 g. (66%).

II. According to Goehring and Niedenzu,  $PS(NH_2)_3$  can also be produced through direct reaction of  $PSCl_3$  with liquid  $NH_3$  in the manner described in method II for  $PO(NH_2)_3$ , using the apparatus shown in Fig. 199. The Klement and Koch procedure (diethylamine) is used to separate the products. The yield is 96%.
PROPERTIES:

Colorless rhombic crystals. Insoluble in ethanol, readily soluble in methanol, very readily in water. Heating with sodium hydroxide produces sodium diamidomonothiophosphate:  $PS(NH_2)_3 +$ NaOH = NaPOS(NH<sub>2</sub>)<sub>2</sub> + NH<sub>3</sub>. Forms diammonium hydrogen monothiophosphate on standing in moist air for several weeks:  $PS(NH_2)_3 +$  $3H_2O = (NH_4)_2$  HPO<sub>3</sub>S + NH<sub>3</sub>. Therefore, it must be sealed in glass for storing.

**REFERENCES:** 

- I. R. Klement and O. Koch. Chem. Ber. 87, 333 (1954).
- II. M. Goehring and K. Niedenzu. Chem. Ber. 89, 1768 (1956).

# Pyrophosphoryl Tetramide $P_2O_3(NH_2)_4$

Pyrophosphoryl tetramide is produced from pyrophosphoryl tetrachloride and  $NH_3$  in chloroform solution:

 $\begin{array}{c} P_2O_3Cl_4 + 8\,NH_3 = P_2O_3(NH_2)_4 + 4\,NH_4Cl\\ 251.8 & 136.2 & 174.05 & 214.0 \end{array}$ 

When the reaction product is treated with liquid  $\rm NH_3$ , pure  $\rm P_2O_3(\rm NH_2)_4$  is obtained (Klement and Benek, Goehring and Niedenzu). I. The apparatus of Fig. 198 is used. A solution of 8 g. of  $\rm P_2O_3Cl_4$  in 20 ml. of CHCl<sub>3</sub> is added in drops, with stirring and cooling, to a solution of  $\rm NH_3$  in 750 g. of dry CHCl<sub>3</sub>. The procedure described above (p. 584) is then followed and the reaction product is suction-filtered in the absence of air. The product is washed with dry CHCl<sub>3</sub> and is dried in vacuum. The apparatus shown in Fig. 71, p. 88, may be used for extraction with liquid  $\rm NH_3$ . The yield is 75%.

II. As in the method of Goehring and Niedenzu, 25 g. of  $P_2O_3Cl_4$  is treated with 150 ml. of dry liquid NH<sub>3</sub> in the apparatus of Fig. 199. Slow dropwise addition of  $P_2O_3Cl_4$  is necessary as a water insoluble substance will be otherwise formed. The product mixture is worked up in the same apparatus. The yield is 89%.

PROPERTIES:

Colorless crystals, readily soluble in water, giving a weakly acidic reaction. Diammonium diamidopyrophosphate is formed on exposure to moist air and solution in water:

$$(NH_2)_2 P(O) - O - P(O)(NH_2)_2 + 2 H_2 O = NH_4 O_2 P(NH_2) - O - (NH_2) PO_2 NH_4.$$

REFERENCES:

R. Klement and L. Benek. Z. anorg. allg. Chem. <u>287</u>, 12 (1956). M. Goehring and K. Niedenzu. Chem. Ber. <u>89</u>, 1771 (1956).

## Tetrasodium Imidodiphosphate

## $Na_4P_2O_6NH \cdot 10 H_2O$

When anhydrous disodium monoamidophosphate is heated in vacuum to  $210^{\circ}$ C, anhydrous tetrasodium imidodiphosphate is formed (Klement and Biberacher):

 $\frac{2 \text{ Na}_{2} \text{PO}_{3} \text{NH}_{2} = \text{Na}_{4} \text{P}_{2} \text{O}_{6} (\text{NH}) + \text{NH}_{3}}{282.0} \\ \frac{264.9}{17.0}$ 

For a good yield, the Na  $_{2}PO_{3}NH_{2}$  must be completely anhydrous. The freshly produced salt containing water of crystallization (see p. 581) is first stored in a vacuum desiccator over NaOH or CaCl<sub>2</sub> at room temperature for two days. Then the CaCl<sub>2</sub> is replaced with  $P_{2}O_{5}$  and the desiccator is sealed with a lubricant that is little affected by temperature changes, such as a silicone grease. The desiccator is evacuated and placed in an oven, which is then heated to  $70^{\circ}C$  for six hours. The desiccator is left to stand for three days and the  $P_{2}O_{5}$  is replenished as necessary. If the desiccator is tightly sealed, a completely dry substance is obtained.

Absolutely anhydrous  $Na_2PO_3NH_2$  (10-15 g.) is weighed into a 50-ml. ground-joint flask. The flask is connected to a highvacuum system (oil pump) with a stopcock and a trap with solid NaOH interposed. The flask is evacuated and gradually heated to  $80^{\circ}C$ . This temperature is maintained for six hours to remove any residual water. The temperature is then raised; the rising pressure indicates liberation of  $NH_3$ . Evacuation is repeated more frequently and the temperature is raised to  $210^{\circ}C$ . Gas evolution gradually subsides and ends after seven days. To purify the crude  $Na_4P_2O_6NH$ , a 10-g. sample is gradually added to 100 ml. of icecold 0.1N NaOH. The clear solution, heated to about  $10-15^{\circ}C$  by the exothermic hydration, is immediately fractionally precipitated with ice-cold acetone, as shown in the scheme below:



Fraction 1 a' is discarded and the three fractions 1 b", 2 a' and 2 b are combined. They contain pure  $Na_4P_2O_6NH \cdot 10H_2O$ . The crystals are washed with acetone and dried in air.

**PROPERTIES:** 

Formula weight  $(Na_2O_3P-NH-PO_3Na_2 \cdot 10H_2O)$  445.13. Colorless crystals, readily soluble in water. Isomorphous with  $Na_4P_2O_7$  $10H_2O$ . A 0.1 M aqueous solution has a pH of 11. On boiling, decomposition to monoamidophosphate and orthophosphate takes place:  $Na_4PO_8NH + H_2O = Na_2PO_3NH_2 + Na_2HPO_4$ . Orthophosphate is quantitatively formed in acid solution. Heating for seven days in vacuum at 450°C produces sodium nitrilotriphosphate:

$$3 \text{ Na}_4 P_2 O_6 \text{ NH} = 2 \text{ N}(PO_3 \text{ Na}_2)_3 + \text{ NH}_3.$$

REFERENCE:

R. Klement and G. Biberacher. Z. anorg. allg. Chem. 283, 246 (1956).

## SECTION 10

## Arsenic, Antimony, Bismuth P. W. SCHENK

## Arsenic

## As

VERY PURE ARSENIC

Very pure, antimony-free arsenic is produced from magnesium ammonium arsenate. A solution of arsenic acid is mixed with chemically pure MgCl<sub>2</sub> and ammonia is slowly added. The initial precipitate is discarded and the middle fractions are collected. Then the material is repeatedly dissolved and precipitated in a similar fashion, the number of such treatments depending upon the purity of the starting materials. In each case, the first and last fractions are discarded. After final washing, the material is dried. The magnesium ammonium arsenate thus obtained is mixed with sugar charcoal; pure As is sublimed from the mixture by heating, preferably in vacuum.

De Pasillé recommends that ammonium arsenate be reduced in an ammonia stream at  $1000^{\circ}$ C. According to Geach, very pure As can also be obtained by reduction of chemically pure As<sub>2</sub>O<sub>3</sub> with Zr metal and subsequent sublimation. Impurities other than Sb can be removed by sublimation in vacuum (Geach).

**PROPERTIES:** 

M.p.  $817^{\circ}$  (under pressure), b.p.  $633^{\circ}$ ; d. 5.78. Rhombohedral crystals with a gray metallic luster. Brittle. Hardness (Mohs) 3.5.

**REFERENCES:** 

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## G. A. Geach, R. A. Jeffery and R. J. Shelton. J. Chem. Soc. (London) 1950, 1207. [For preparation of single crystals, see P. W. Bridgman, Proc. Am. Acad. <u>68</u>, 27-93, 39 (1932-33).]

## YELLOW ARSENIC

Yellow As is formed when the vapor is cooled suddenly. In the Erdmann and Unruh method, the As vapor is added directly to CS<sub>2</sub>, which serves as the solvent, because the yellow form converts to the gray with exceptional ease, particularly in the light. Figure 200 illustrates the apparatus employed. The aluminum tube a (wall thickness 3 mm., inside diameter 20 mm., length 1 m.) is wrapped with a 2-mm. thickness of asbestos paper (b), which is cemented in place with water glass. A closely fitting, 40-cm. iron tube e is pushed over the tube. The end of the Al tube projecting 50 cm. is closed with a one-hole cork and connected to a CO<sub>2</sub> source via a wash bottle containing H<sub>2</sub>SO<sub>4</sub>. The other end of the Al tube is reinforced with asbestos and water glass to provide a good seal with the conical aperture of tube f, even though that end of the tube is very hot during operation because of the necessity of preventing condensation and consequent plugging with As. Receiver g is a 12-cm. I.D. sphere with U tube f attached. Dry CO<sub>2</sub> at about  $-20^{\circ}$ C is introduced through a tube adjacent to the thermometer. The receiver stands in ice-water bath and holds about 300 ml. of CS<sub>2</sub>. A second and a third receiver, identical but containing only about 100-120 ml. of CS2 each, are connected in line with the first. All three vessels are placed in the ice-water bath, which is covered to exclude light. The level of ice water must extend above the conical connection of tube f. The Al tube is charged with pure As



Fig. 200. Preparation of yellow arsenic. a) Aluminum tube; b) asbestos insulation; d) asbestos paper heat insulation; f) inlet tube; g) receiver.

and brought to low red heat by a gas flame or an electric furnace. The thin oxide layer on the As is immediately removed by heating the material in a stream of  $CO_2$ . Then an additional  $CO_2$  stream is admitted through the aluminum tube. Thus, the As vapor which distills over is cooled with the  $-20^{\circ}C$   $CO_2$  and is immediately absorbed in the cold  $CS_2$ . The solution is filtered and concentrated on a water bath, to about half the original volume. It is then evaporated to saturation in a large diameter test tube. Finally, it is cooled to  $-70^{\circ}C$  in a Dry Ice-acetone bath, whereupon the As precipitates. The supernatant liquid is decanted and the As is dried in vacuum. All the operations, particularly those with the concentrated solutions, must be performed as far as possible in the absence of light. About 12 g. of yellow arsenic is obtained from the solution in the first receiver and another 4 g. from the second. There is none in the third.

#### PROPERTIES:

Yellow powder, converting to gray As in less than one minute when exposed to sunlight. Soluble in  $CS_2$ . Cubic crystals. d. (20°C) 1.97.

#### Solubility in 100 ml. of $CS_2$

Temperature, °C:	+ 46	+ 18 to 20	+ 12	0	-15	-60 to -80
Yellow As. g.:	11	7.5—8	5.5 - 6	3.8-4	2 - 2.5	0.8-1.0

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## Arsine

#### AsH<sub>3</sub>

I.	$Na_sAs + 3$	$3 NH_4Br =$	3  NaBr +	$3 NH_3 +$	AsH <sub>3</sub>
	143.9	293.9	308.7	51.1	22.4 l (77.9)

The reaction is run in liquid  $NH_3$ , using the reactor illustrated in Fig. 201. First the required amount of Na is placed in a and dry  $NH_3$  is condensed over it. The required amount of powered As is then added. By turning and tapping the bulb, dry  $NH_4$  Br, which has been weighed into the flask at the right, is added to a. The evolving gas is scrubbed with water, dried with  $P_2O_5$ , and condensed in a trap cooled with liquid nitrogen. It can be purified by vacuum



Fig. 201. Preparation

of arsine.

distillation in the apparatus used for  $SbH_3$  (p. 599, Fig. 203, or Part I, p. 67).

II. 
$$As_2Zn_3 + 3 H_2SO_4 = 3 ZnSO_4 + 2 AsH_3$$
  
345.9 294.2 484.3 44.8 l (155.8)

To obtain the required zinc alloy, an intermediate alloy is first prepared by fusing the quantity of Zn computed from

$$\begin{array}{l} 2\,A_{\text{S}}\,+\,3\,\text{Zn}=A_{\text{S}_2}\text{Zn}_3\\ 149.8 \quad 196.1 \quad 345.9 \end{array}$$

in a closed crucible, to which a portion of the As is added. After cooling, the alloy is pulverized, the rest of the As is added, and the fusion in the closed crucible is repeated. After cooling and grinding, the alloy is placed in a reaction flask equipped with a dropping funnel and a ground joint. The air is displaced with a stream of  $H_{24}$ .

and boiled 30% H<sub>2</sub>SO<sub>4</sub> is then added drop-by-drop. The gas thus generated contains only 0.5% impurities. It is scrubbed with KOH solution, dried with solid KOH and P<sub>2</sub>O<sub>5</sub>, and purified by condensation and vacuum distillation at low temperature.

III. Arsine can also be prepared in the familiar manner involving addition of a solution of  $As_2O_3$  in HCl to zinc granules and dilute  $H_2SO_4$ . However, the gas thus generated contains no more than 25%  $AsH_3$ .

IV. In the Nast method, an Mg-Al-As alloy is decomposed with very dilute sulfuric acid. Fractionation of the resulting gases in a Stock vacuum apparatus also permits isolation (even though in a very small quantity) of As<sub>2</sub>H<sub>4</sub> (diarsine), an analog of N<sub>2</sub>H<sub>4</sub> and P<sub>2</sub>H<sub>4</sub>. V. According to Stone and Burg, treatment of AsCl<sub>3</sub> with LiAlH<sub>4</sub> in absolute ether gives AsH<sub>3</sub> in a 70% yield.

## PROPERTIES:

M.p. -113°C, b.p. -58.5°. Colorless gas and liquid. Exceptionally poisonous. Sensitive to O<sub>2</sub>. Unstable on porous surfaces.

## **REFERENCES:**

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## Sodium Dihydrogen Arsenide

## NaAsH<sub>2</sub>

$$\begin{array}{rl} N_{a} + A_{s}H_{s} = N_{a}A_{s}H_{2} + \frac{1}{2}H_{2} \\ 23.0 & 77.9 & 99.9 & 11.2 \, l \end{array}$$

Gaseous arsine is introduced into a solution of sodium in liquid  $NH_3$  at  $-78^{\circ}C$ . The arsine apparatus described in Fig. 201 is employed. Approximately 1 g. of sodium is placed in the side bulb and about 50 ml. of  $NH_3$  is condensed in  $\alpha$ . Then the Na is added to the  $NH_3$  by rotation of the bulb, and dissolved while the flask is cooled with Dry Ice. A stream of  $AsH_3$  is then introduced, while the flask is still in the cold bath. The solution, initially dark blue, becomes lighter and finally turns yellow. When the color changes, the  $NH_3$  is evacuated. The residue is  $NH_3$ -free  $NASH_2$  which appears on the reactor walls as pale-yellow crystals.

Potassium dihydrogen arsenide can be produced in a similar fashion.

Other preparative methods:

$$(C_{6}H_{5})_{3}CNa + AsH_{3} = (C_{6}H_{5})_{3}CH + NaAsH_{2}$$
  
R + 23.0 77.9 99.9

#### PROPERTIES:

Decomposes rapidly in air, generating heat and leaving a brown residue of unknown composition. Stable in vacuum, in the absence of air. Hydrolyzes in  $H_2O$ , forming AsH<sub>3</sub>.

See also the preparation and properties of the analog NaPH<sub>2</sub>.

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- W. C. Johnson and A. Pechukas. J. Amer. Chem. Soc. <u>59</u>, 2068 (1937).
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## **Arsenic Trichloride**

## AsCl<sub>3</sub>

I.

$$2 \text{ As} + 3 \text{ Cl}_2 = 2 \text{ AsCl}_3$$
  
149.8 212.7 362.5

Pulverized As is placed in a retort with a filler tube or in a boat inserted into a Pyrex tube. Dry  $Cl_2$  is passed over the material. The As usually ignites and burns in the  $Cl_2$  stream. If necessary, the reaction may be initiated by slight heating. A long cooling tube attached to the reaction tube (or to the neck of the retort) delivers the product to a distillation flask serving as receiver. At the end, a small quantity of As powder is added to the distillation flask to bind the dissolved Cl, and the contents are distilled. The material fractionated over As powder is completely pure.

II.  $4 \operatorname{As}_2 \operatorname{O}_3 + 3 \operatorname{S}_2 \operatorname{Cl}_2 + 9 \operatorname{Cl}_2 = 8 \operatorname{As} \operatorname{Cl}_3 + 6 \operatorname{SO}_2$  $791.3 \quad 405.1 \quad 638.2 \quad 1450.2 \quad 384.4$ 

A mixture of 140 g. of  $As_2O_3$  and 100 g. of  $S_2Cl_2$  is prepared in a flask, and dry  $Cl_2$  is introduced. When most of the material has reacted, another 60 g. of  $As_2O_3$  is added, and additional  $Cl_2$  is introduced. Finally, the product is distilled directly from the flask, in the presence of some metallic arsenic.

III.  $\begin{array}{c} A_{S_2}O_3 + 6 \ HCl = 2 \ A_SCl_3 + 3 \ H_2O \\ 197.8 \ 218.8 \ 362.5 \ 54.0 \end{array}$ 

The  $As_2O_3$  is distilled from concentrated hydrochloric acid in a stream of HCl.

#### PROPERTIES:

Formula weight 181.28. Colorless, oily liquid. Fumes in air. Extremely toxic (skin poison). Solidifies at  $-16.2^{\circ}$ C to colorless crystals with a pearly sheen. B.p. 130.2°C; d. 2.2. Soluble in H<sub>2</sub>O and HCl. Dissolves alkali iodides, sulfur, phosphorus, and oils.

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## Arsenic Tribromide

## AsBr<sub>3</sub>

 $2\,\mathrm{As} + 3\,\mathrm{Br}_2 = 2\,\mathrm{AsBr}_3$ 

629.3

479.5

149.8

I.

A boat is placed in a Pyrex tube and filled with As powder. A stream of dry  $N_2$ , saturated with  $Br_2$  vapor by passage through a  $Br_2$ -filled wash bottle, is passed over the As. The tube is inclined toward the receiver and connected with the latter by means of an asbestos-paper-sealed adapter. The tube is heated until the onset of reaction. The product is distilled from the receiver over As powder.

II.

$$2 \text{As}_2\text{O}_3 + 3 \text{S} + 6 \text{Br}_2 = 4 \text{As}\text{Br}_3 + 3 \text{SO}_2$$
  
395.6 96.2 959.0 1258.6 192.2

A mixture of 26.5 g. of  $As_2O_3$ , 6 g. of sulfur, and 64 g. of  $Br_2$  is heated in a 300-ml. flask in a metal bath. After about seven hours, the rising vapors are no longer colored brown with bromine. A preheated suction filter is employed for filtering. The product is very pure. Further purification involves distillation over As as in method I.

PROPERTIES:

Formula weight 314.66. Colorless, deliquescent, rhombic prisms. M.p. 31.2°C, b.p. 221°C; d. ${}^{1_{4}5}_{4}$  3.66. Fumes in humid air and is hydrolyzed by water.

**REFERENCES:** 

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#### Arsenic Triiodide

AsI<sub>3</sub>

$$A_{s} + \frac{3}{2}I_{2} = A_{s}I_{3}$$
  
74.9 380.7 455.6

I. A CS<sub>2</sub> or ether solution of  $I_2$  is refluxed with excess As powder until the iodine color disappears. This is followed by rapid filtration. The product is then allowed to crystallize. Concentration of the solution yields additional crystals. The product is recrystallized from CS<sub>2</sub> or ether. Other preparative methods:

II.

A solution of 10 g. of KI in 10 ml. of  $H_2O$  is added to a hot solution of 2 g. of  $As_2O_3$  in 30 ml. of hydrochloric acid (d. 1.19), whereupon a yellow crystalline powder precipitates. This is suctionfiltered and washed with concentrated hydrochloric acid (d. 1.12) until no KCl residue is seen when the wash liquid is evaporated on a watch glass. The product can be further purified by recrystallization as described above.

III. In the Oddo method, the product is obtained from  $As_2O_3$ , iodine, and sulfur, in a procedure identical to that described for  $AsBr_3$ . However, 102 g. of iodine is used in the present case. After heating for 14 hours, the product can be purified by recrystallization as described above.

## PROPERTIES:

Lustrous, trigonal red leaves or platelets. Rhombohedral. M.p. 141.8°C, b.p. 403°C; d. $_{4}^{25}$  4.688. Soluble in H<sub>2</sub>O, ether, CS<sub>2</sub>, xylene, and various other organic solvents. Sparingly soluble in strong HCl. In air, gradually decomposes into AS<sub>2</sub>O<sub>3</sub> and iodine.

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- L. Vegard. Skr. Acad. Oslo (1947) No. 2, 1.
- W. Biltz and A. Sapper. Z. anorg. allg. Chem. 203, 277 (1932).

## Arsenic Diiodide

AsI2

 $\begin{array}{l} As + I_2 = AsI_2 \\ 74.9 \quad 253.8 \quad 328.7 \end{array}$ 

A mixture of 20 g. of  $I_2$  and 20 g. of powdered As is placed in a tube, and all air is displaced by repeated evacuation and filling

598

with  $CO_2$ . The tube is sealed in a stream of  $CO_2$ . The tube is then heated in an inclined position for seven hours at 240°C, and permitted to cool in the same position to enable the product to gather and solidify at one end. To recrystallize the product, the section containing the product is broken off. It is immediately placed in flask a of the apparatus illustrated in Fig. 202, prefilled with airfree,  $P_2O_5$ -dried  $CO_2$  from *b*. The flask capacity is 300-400 ml. and it contains about 150 ml. of dry xylene. Stopper *c* is then pushed firmly into place while d is left loose. The xylene is then boiled (continued passage of CO<sub>2</sub>) until the AsI<sub>2</sub> dissolves. After that, d is tightened in place, c is removed, and adapter e, containing a hard folded filter paper f, is put in its place. Flask h is then placed over  $\boldsymbol{e}$  while a constant flow of CO<sub>2</sub> is maintained. After h has been filled with CO<sub>2</sub> (in about 5 minutes) the solution is transferred into h via filter f by slowly tilting the apparatus. The solution must remain boiling hot. The  $CO_2$  inlet is then transferred to g and the stopper connecting h to e may be removed. The product is permitted to cool in the CO<sub>2</sub> stream; the AsI<sub>2</sub> precipitates as long red prisms.



Fig. 202. Purification of arsenic diiodide.

To separate the AsI<sub>2</sub>, a fairly coarse fritted glass filter i, attached to suction flask k, is placed over the neck of flask h. Carbon dioxide is introduced through a tube in the stopper of k. When the whole system has been filled with CO<sub>2</sub> via g, the apparatus is turned upside down and the AsI<sub>2</sub> drops onto the fritted filter. It is then suction dried in a stream of CO<sub>2</sub>. Removal from the filter must also be done in a stream of CO<sub>2</sub> and in the absence of moisture. The aspirator is turned off, the system is filled with CO<sub>2</sub>, and flask h is removed quickly and replaced by an adapter l which fits the fritted filter. The adapter is sealed to the filter with large-diameter rubber hose (e.g., bicycle inner tube). Then CO<sub>2</sub> passage from below is resumed. Several previously prepared storage tubes may now be attached at the other end of the adapter. These are filled with  $CO_2$ , the  $AsI_2$  is poured in, and the tubes are sealed with a torch.

PROPERTIES:

Red prisms. M.p. 128-130°C. Exceptionally unstable in air. Disproportionates in water to As and AsI<sub>2</sub>.

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## Diarsenic Trioxide

As<sub>2</sub>O<sub>3</sub> (As<sub>4</sub>O<sub>6</sub>)

 $2 \operatorname{AsCl}_{3} + 3 \operatorname{H}_{2}O = \operatorname{As}_{2}O_{3} + 6 \operatorname{HCl}_{362.5}$ 54.0 197.8 218.8

To obtain a pure product, 150 g. of  $As_2O_3$  is treated with concentrated HCl in the apparatus shown in Fig. 231, and  $AsCl_3$  is slowly distilled in a stream of HCl. To remove SbCl<sub>3</sub> from the oily layer in the first condensation trap, the material is shaken repeatedly with 2/3 its volume of concentrated hydrochloric acid until the aqueous layer is free of Sb (tested by the method of Foulk and Horton, i.e., distilling off the bulk of dissolved  $AsCl_3$  and precipitating the strongly acid solution with  $H_2S$ . The precipitate is filtered off, the solution diluted, and the  $H_2S$  bubbling is resumed; Sb precipitates as an orange-red solid. If SbCl<sub>3</sub> is present only in traces, the precipitate appears only after one or two days of standing in an atmosphere of  $H_2S$ ). The pure  $AsCl_3$  is hydrolyzed in vigorously stirred boiling water and  $As_2O_3$  crystallizes out on cooling.

PROPERTIES:

White powder. There are two crystalline modifications, monoclinic claudetite (m.p.  $313^{\circ}$ C) and cubic arsenolite (m.p.  $275^{\circ}$ C). A cold solution of As<sub>2</sub>O<sub>3</sub> in hydrochloric acid always separates arsenolite. Claudetite can be obtained be heating moist arsenolite for several days at 100 to  $200^{\circ}$ C. In addition, there is a vitreous form of  $As_2O_3$ . In the vapor and in nitrobenzene the formula is  $As_4O_6$ . The structure is analogous to that of  $P_4O_6$ . B.p. 465°C; sublimes at 195°C in a stream of  $N_2$ .

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## Orthoarsenic Acid

## H<sub>3</sub>AsO<sub>4</sub>

$$\begin{array}{l} \text{As}_2\text{O}_3 + 2 \text{ HNO}_3 + 3 \text{ H}_2\text{O} = \text{N}_2\text{O}_3 + 2 \text{ H}_3\text{As}\text{O}_4 \cdot {}^{1}\!/_2 \text{ H}_2\text{O} \\ 197.8 & 126.0 & 54.0 & 76.0 & 301.8 \end{array}$$

Arsenic trioxide (100 g.) is placed in a ground joint flask and 100 ml. of concentrated HNO<sub>3</sub> (d. 1.38) is slowly added from a dropping funnel. The mixture is then heated. The nitric oxides are passed through concentrated H<sub>2</sub>SO<sub>4</sub> to obtain nitrosyl hydrogen sulfate. When the evolution of nitrogen oxides ceases, the supernatant liquid is decanted from the undissolved material and evaporated to dryness. The residue is taken up in some H<sub>2</sub>O and filtered through a glass frit, and the solution is evaporated until a thermometer in the liquid reads 130°C. The solution, which has a honeylike consistency when cold, is left to crystallize in a refrigerator (in a desiccator over H<sub>2</sub>SO<sub>4</sub>). Beautiful transparent crystals of H<sub>3</sub>AsO<sub>4</sub> · 1/2 H<sub>2</sub>O precipitate. If the crystallization is done at -30°C, the hydrate H<sub>7</sub>AsO<sub>6</sub> may be formed, but only if the solution is held for a few days at this temperature. If the solution is evaporated until a b.p. of 175°C is reached, As<sub>2</sub>O<sub>5</sub> · 5/3 H<sub>2</sub>O precipitates.

PROPERTIES:

All the hydrates lose their water on heating above 250-300  $^{\circ}C$  and convert to As  $_{2}O_{\text{5}}$ .

 $H_3AsO_4 \cdot 1/2$   $H_2O$ : transparent, large, hygroscopic crystals. As  $_2O_5 \cdot 5/3$   $H_2O$ : dull, granular crystalline mass.

REFERENCES:

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## Sodium Dihydrogen Orthoarsenate

## NaH<sub>2</sub>AsO<sub>4</sub> · H<sub>2</sub>O

 $\begin{array}{c} H_{3}AsO_{4} + NaOH = NaH_{2}AsO_{4} \cdot H_{2}O \\ 141.9 & 40.0 & 181.9 \end{array}$ 

A 5N solution of  $H_3AsO_4$  is neutralized with caustic soda to the methyl orange end point and then evaporated until crystallization begins. The crude salt which precipitates on cooling is dissolved in water to make a saturated solution at 100°C (75.3 g./100 g.  $H_2O$ ), and is then cooled to 0°C with stirring. The fine, crystalline precipitate is suction-filtered, washed with some ice water, and dried over  $P_2O_8$ .

PROPERTIES:

White crystalline powder. d 2.53. Loses water of crystallization when warmed and passes through the stages  $NaH_2AsO_4 \rightarrow Na_2H_2As_2O_7 \rightarrow Na_3H_2As_3O_{10}$ , finally converting to  $NaAsO_3$  above 230°C.

REFERENCE

E. Thilo and J. Plaetschke. Z. anorg. Chem. 260, 315 (1949).

## Ammonium Orthoarsenate

 $(NH_4)_3AsO_4 \cdot 3H_2O$ 

 $\begin{array}{rl} H_{3}AsO_{4} + 3 NH_{3} + 3 H_{2}O = (NH_{4})_{3}AsO_{4} \cdot 3 H_{2}O \\ 141.9 & 51.1 & 54.0 & 247.1 \end{array}$ 

A saturated aqueous orthoarsenic acid solution is saturated with gaseous NH<sub>3</sub>. Beautiful crystals of ammonium orthoarsenate precipitate immediately.

PROPERTIES:

Rhombic crystalline lamellae. Loses  $NH_3$  and  $H_2O$  in air and converts to the monohydrogen salt. On heating, the aqueous solution loses so much  $NH_3$  that the solution corresponds at the end to the dihydrogen salt.

**REFERENCES:** 

- H. Salkowski. J. prakt. Chem. 104, 132 (1868).
- C. Matignon and A. de Passille. Comptes Rendus Hebd. Séances Acad. Sci. 198, 779 (1934); 200, 1854 (1935).

## Tetraarsenic Tetrasulfide

 $As_4S_4$ 

 $As + S = \frac{1}{4} AsS$ 74.9 32.1 107.0

Equimolar quantities of As and S are placed in a thick-walled, sealed tube and fused together at  $500-600^{\circ}$ C in a N<sub>2</sub> atmosphere. The product is sublimed in vacuum.

**PROPERTIES:** 

Deep red, lustrous monoclinic crystals. Vapor density measurements at 550°C indicate As  $_4$ S  $_4$ . M.p. 320°C, b.p. 565°C; d 3.5. Occurs in nature as realgar.

**REFERENCE:** 

E. V. Britzke and A. F. Kapustinski. Z. anorg. allg. Chem. 205, 95 (1932).

## Diarsenic Pentasulfide

#### $As_2S_5$

 $\begin{array}{r} 2\,\mathrm{H_{3}A_{5}O_{4}} + 5\,\mathrm{H_{2}S} = \mathrm{As_{2}S_{5}} + 8\,\mathrm{H_{2}O} \\ 283.9 & 170.4 & 310.2 & 144.1 \end{array}$ 

Pure As  $_2$ S  $_5$  can be obtained by precipitation of orthoarsenic acid solutions only if the aqueous acid is treated with twice its volume of concentrated (d 1.19) hydrochloric acid, and a fast stream of H<sub>2</sub>S is bubbled through the ice-cooled solution for about an hour. The flask must be cooled even during the HCl addition and no temperature rise should be allowed. The precipitate is washed with water and alcohol and dried at 100°C.

**PROPERTIES:** 

Somewhat brighter yellow than the trisulfide. Decomposes into  $As_2O_3$ , S, and  $As_2S_3$  when boiled with  $H_2O$ .

**REFERENCES:** 

F. Neher. Z. anal. Chem. <u>32</u>, 45 (1893).

- W. Foster. J. Amer. Chem. Soc. <u>38</u>, 52 (1916).
- F. Foerster. Z. anorg. allg. Chem. 188, 90 (1930).

## Ammonium Thioarsenate

(NH<sub>4</sub>)<sub>3</sub>AsS<sub>4</sub>

 $\begin{array}{l} As_2S_5 + 3\,(NH_4)_2S = 2\,(NH_4)_3AsS_4\\ 310.2 \quad 204.5 \quad 514.6 \end{array}$ 

A solution of As  ${}_{2}S_{5}$  in excess (NH<sub>4</sub>)  ${}_{2}S$  is prepared with moderate heating and thoroughly extracted with hot alcohol. Cooling results in precipitation of colorless crystals of the salt. After suction filtering, the product is washed with alcohol.

PROPERTIES:

Formula weight 257.29. Prisms. In air, the surface yellows. Melts on heating and decomposes to  $As_2S_3$ , S, and ammonium sulfide.

**REFERENCE:** 

W. P. Bloxam. J. Chem. Soc. (London) 67, 277 (1895).

## Sodium Thioarsenate

Na<sub>3</sub>AsS<sub>4</sub> · 8 H<sub>2</sub>O

 $\begin{array}{c} 3\,\mathrm{Na_2S} + \mathrm{As_2S_3} + 2\,\mathrm{S} = 2\,\mathrm{Na_3AsS_4} \\ 234.1 & 246.0 & 64.1 & 544.3 \end{array}$ 

A solution of 20 g. of  $As_2O_3$  in hot NaOH is strongly acidified with HCl. Then  $As_2S_3$  is precipitated by bubbling  $H_2S$  through the hot solution. Filtering and thorough washing with dilute HCl follows. One half of a solution of 24 g. of NaOH in 100 ml. of water is saturated with  $H_2S$  and mixed with the other half. The thus obtained Na<sub>2</sub>S solution is used to dissolve 6.4 g. of sulfur and the  $As_2S_3$ . The mixture is evaporated on a steam bath until a crystalline surface layer forms. It is then left to crystallize in the cold.

PROPERTIES:

Formula weight 416.275. Monoclinic prisms, colorless to pale yellow. Readily soluble in  $H_2O$  and stable in air. Acidification of the solution results in precipitation of copious quantities of  $As_2S_5$ .

**REFERENCES:** 

R. Fresenius, Z. analyt. Chem. <u>1</u>, 192 (1862). McCay. Z. analyt. Chem. <u>34</u>, 725 (1895).

## Sodium Monothioorthoarsenate

Na<sub>3</sub>AsO<sub>3</sub>S · 12 H<sub>2</sub>O

 $\begin{array}{ccc} 6 \text{ NaOH} + \text{ As}_2\text{O}_3 + 2 \text{ S} = 2 \text{ Na}_3\text{AsO}_3\text{S} + 3 \text{ H}_2\text{O} \\ 240.0 & 197.8 & 64.1 & 447.9 & 54.0 \end{array}$ 

Arsenic trioxide (20 g.) is dissolved in a solution of 24 g. of NaOH in 100 ml. of  $H_2O$  and the mixture is boiled with 6.5 g. of S for half an hour. It is filtered hot and evaporated on a steam bath until crystallization. It is then permitted to cool and the crystals are suction filtered.

**PROPERTIES:** 

Formula weight 440.14. Colorless, rhombic columns, readily soluble in water. Effloresces in air.

REFERENCES:

R. F. Weinland and O. Rumpf. Ber. dtsch. Chem. Ges. <u>29</u>, 1009 (1896); Z. anorg. allg. Chem. 14, 42 (1897).

## Sodium Dithioorthoarsenate

 $Na_3AsO_2S_2 \cdot 11 H_2O$ 

 $\begin{array}{c} 6 \text{ NaOH} + \text{As}_2\text{S}_3 + 2 \text{ S} = 2 \text{ Na}_3\text{AsO}_2\text{S}_2 + \text{H}_2\text{S} + 2 \text{ H}_2\text{O} \\ 240.0 \quad 246.0 \quad 64.1 \quad 480.0 \quad 34.1 \quad 36.0 \end{array}$ 

A mixture of 24.6 g. of moist, freshly precipitated As  $_2S_3$  and 6.4 g. of finely ground sulfur is prepared and a solution of 24 g. of NaOH in 200 ml. of water added. This new mixture is allowed to stand for 12 hours, with frequent agitation, until the bulk of the sulfur has dissolved. It is then filtered, and the filtrate is evaporated until crystallization, at which point it is allowed to cool. The crystallized substance consists of very pure Na<sub>3</sub>AsO<sub>2</sub>S<sub>2</sub> · 11 H<sub>2</sub>O. Additional product is obtained from the mother liquor; it is, however, contaminated with thio- and monothioarsenates.

**PROPERTIES:** 

Formula weight 438.19. Colorless, rhombic crystals, readily soluble in water.

**REFERENCE:** 

R. F. Weinland and P. Lehmann, Z. anorg. allg. Chem. <u>26</u>, 340 (1901).

## Antimony

#### Sb

Chemically pure Sb can be produced by reduction of  ${\rm Sb}_2O_5$  with hydrogen or, more simply, with KCN. Equal parts of the dried oxide and chemically pure KCN are mixed and the mixture is fused in a porcelain crucible. After cooling, the block of metal is freed of surface impurities by boiling with water and remelted. The starting  ${\rm Sb}_2O_5$  is best purified by conversion to  ${\rm HSbCl}_6$  through repeated recrystallization from pure, concentrated hydrochloric acid (Groschuff method). Lautié recommends vacuum distillation of the metal at 800°C for final purification.

PROPERTIES:

M.p.  $630^{\circ}$ C, b.p.  $1645^{\circ}$ C; d 6.684. Brittle, lustrous, bluishwhite metal, can be pulverized. Insoluble in HF, HCl, and H<sub>2</sub>SO<sub>4</sub>. Soluble in HNO<sub>3</sub>-tartaric acid solution and in aqua regia. Rhombohedral crystals.

**REFERENCES:** 

- E. Groschuff. Z. anorg. allg. Chem. 103, 164 (1918).
- H. H. Willard and R. K. McAlpine. J. Amer. Chem. Soc. <u>43</u>, 801 (1921).
- O. Hönigschmid, E. Zintl and M. Linhard. Z. anorg. allg. Chem. <u>136</u>, 264 (1924).
- R. Lautié. Bull. Soc. Chim. France (5) 14, 975 (1947).

## Stibine (Antimony Hydride)

SbH₃

I.

 $\begin{array}{c} Mg_3Sb_2 + 6 \ HCl = 2 \ SbH_3 + 3 \ MgCl_2 \\ 316.5 \ 218.8 \ 249.6 \ 285.7 \end{array}$ 

A mixture of 20 g. of finely pulverized Sb and 40 g. of Mg powder is placed in a sheet-iron trough (70 cm. long) which is inserted into a 25-mm.-diameter iron tube. The mixture is then heated in a stream of  $H_2$ . When the tube glows, the flame is extinguished and the tube is allowed to cool in the stream of  $H_2$ . The alloy should be sintered but not fused. The tube contents are sieved and the coarse fraction is repulverized. The gray-black powder should pass through a screen with 0.5-mm. openings. In decomposition the powder is added to the acid and not vice versa. Where small quantities are involved, this can be done in the manner described for  $AsH_3$  (p. 593). The acid is placed in flask a of Fig. 203 and Mg-Sb alloy in the side bulb. The powder is added to the flask by turning and tapping the bulb. However, it is better to employ an automatic charging apparatus, illustrated in Fig. 203. The vessel b containing the Sb-Mg powder is fitted on top of flask g. A hard rubber rod d (a thick knitting needle if available) is positioned axially in vessel b and rotating in the mercury seal c. The lower end of the rod is conical, with a few spiral grooves at the bottom. Slow rotation of the rod permits a uniform powder flow into g. The latter contains preboiled, dilute hydrochloric acid (d 1.06) and is cooled with an ice-salt bath. The air is displaced with H<sub>2</sub> prior to addition of the alloy powder. The interior of empty tube e is coated with P<sub>2</sub>O<sub>5</sub> powder, while a glass bead-P<sub>2</sub>O<sub>5</sub> mixture is placed in drying tubes f. The evolving gas passes through e and f for drying and is frozen in trap k, cooled with liquid nitrogen. The product gas contains



Fig. 203. Preparation of stibine. e) Empty tube coated with  $P_2O_5$  powder; f) drying tube filled with  $P_2O_5$  and glass beads; k, l, m, r) condensation vessels; q) Hq vacuum gauge.

about 15% SbH<sub>3</sub>. At the end of the run (SbH<sub>3</sub> evolution is readily controlled by adjustment of the rate of rotation of the feed rod; it continues for a few hours) all the SbH<sub>3</sub> is displaced from g into kwith a stream of H<sub>2</sub>. Capillary *i* is melted and the sealed apparatus is evacuated. The trap contents are melted; a small forerun is distilled into *l*, while the main fraction is collected in *m*. Constrictions *n* and *p* are then melted and the material distilled into ampoules *r*, observing the vacuum on gauge *q*. The distillation temperature is 65-75°C.

Other preparative methods: The electrolytic preparation method of Reismann, Berkenblit, Haase and Gaines uses a Pt anode and a Pt/Ir cathode in an electrolyte consisting of 1.7 liters of  $4N H_2SO_4$ , 80 g. of tartaric acid, and 8 g. of Sb.

**PROPERTIES:** 

Formula weight 124.78. Cubic crystals. The gas decomposes readily, evolving heat (the decomposition may become explosive). Since a flame may not be brought near the SbH<sub>3</sub>, the storage flask may be torch-sealed only if the contents are frozen at liquid N<sub>2</sub> temperature or a long capillary is interposed between the seal point and the flask. M.p.  $-91^{\circ}$ C, b.p.  $-17^{\circ}$ C, d ( $-17^{\circ}$ C) 2.2. Slowly decomposes into the elements just above the b.p. Extremely toxic. Five volumes of H<sub>2</sub>O dissolve one volume of SbH<sub>3</sub>. Cubic crystals.

**REFERENCES:** 

- H. Reckleben and A. Güttig. Z. analyt. Chem. 49, 73 (1910);
- F. Paneth. Z. Elektrochem. 26, 453 (1920);
- H. J. Sand, E. J. Weeks and S. W. Worell. J. Chem. Soc. (London) 123, 456 (1923);
- H. J. S. Weeks. Recueil Trav. Chim. Pays-Bas <u>43</u>, 649 (1924); <u>44</u>, 201, 795 (1925);
- A. Stock and W. Doht. Ber. dtsch. chem. Ges. 35, 2274 (1902);
- G. V. Teal. US Patent 2 391 280 (1945);
- A. A. Durrant, T. G. Pearson and P. L. Robinson. J. Chem. Soc. (London) <u>1934</u>, 733;
- A. Reismann, M. Berkenblit, E. C. Haas and A. Gaines. J. Electrochem. Soc. 101, 387 (1954).

## Antimony (III) Chloride

SbCl<sub>3</sub>

 $2 \text{ Sb} + 3 \text{ Cl}_2 = 2 \text{ SbCl}_3$ 243.5 212.7 456.2

I.

A few pieces of pure Sb are placed in glass tube r (Fig. 204), and a stream of dry  $Cl_2$  is introduced through side arm a of the flask. Tube r rests on an asbestos-lined iron trough inclined slightly toward the flask. Once the reaction has started, it is necessary to add fresh pieces of Sb only from time to time. This is done by removing stopper c, which is only loosely inserted. Plugging does not occur since the chloride is kept fluid in the narrow 8-mm. section of tube r by absorbing  $Cl_2$ . When enough crude chloride is collected in the flask, the  $Cl_2$  stream is interrupted and a few pieces of Sb are placed in the flask. Heat is then applied and finally some more Sb powder is added to eliminate the last of the SbCl<sub>5</sub>. Finally, the SbCl<sub>3</sub> is purified by distillation.

II. A solution of 25 g. of finely pulverized stibnite in 150 ml. of concentrated hydrochloric acid is prepared with heating and filtered



Fig. 204. Preparation of antimony (III) chloride. r) Glass tube, straight section about 60 cm. long, inside diameter 1.5 cm., 0.8 cm. in constricted section. Walls 1.5-2 mm. thick.

after cooling. The operation must be conducted under a hood. The filtrate is fractionated in the presence of 2 g. of Sb, the flask being closed with an asbestos stopper. The condenser is air cooled. The forerun (boiling up to  $120^{\circ}$ C) is discarded. The fraction boiling above  $200^{\circ}$ C is retained. This portion is fractionated once more in the presence of 1 g. of Sb powder, and only the fraction boiling at  $223^{\circ}$ C is collected.



Fig. 205. Purification of antimony (III) chloride by sublimation. t) Drying tube.

A particularly pure preparation can be obtained by resublimation of the SbCl<sub>3</sub>. In this procedure, a two-liter distillation flask k is placed on a steam bath (see Fig. 205) while the upper part of the sphere is cooled with running water. Some SbCl<sub>3</sub> is placed in the flask. When enough fine, long crystals have deposited on the cold upper part and no unsublimed SbCl<sub>3</sub> is left on the bottom, the flask is left to cool without being disturbed. Then the readily detached crystals are transferred (with tapping) into flask v. The entire apparatus must, of course, be carefully dried. PROPERTIES:

Formula weight 228.1. Colorless; when sublimed, long, fine, rhombic crystals. M.p.  $72.9^{\circ}$ C, b.p.  $223.0^{\circ}$ C;  $d_4^{2\circ}$  3.14. Hygroscopic. Fumes in air. Highly corrosive. Soluble in small amounts of H<sub>2</sub>O. In large quantities of H<sub>2</sub>O, hydrolyzes to SbOC1. Soluble in cold alcohol, CS<sub>2</sub>, and in ether; soluble in alkali chloride solutions, forming salts of hexachloroantimony(III) acid. Used for bronzing iron articles. The high cryoscopic constant (18.4) and its ability to dissolve many substances render SbCl<sub>3</sub> suitable for molecular weight determinations.

**REFERENCES:** 

- P. Hensgen. Recueil Trav. Chim. Pays-Bas 9, 301 (1890);
- G. Jander and H. Wendt. Lehrbuch d. analyt. and prap. anorg. Chem. [Analytical and Preparative Inorganic Chemistry] Stuttgart 1954, p. 201;
- J. Kendall, E. D. Crittenden and H. K. Miller. J. Amer. chem. Soc. 45, 967 (1923);
- G. Langguth. Chim. Ind. 25, 22 (1931);
- O. Werner. Z. anorg. allg. Chem. 181, 154 (1929);
- D. I. Zhurevyev. Zhurnal Fiz. Khim. 13, 684 (1939).

## Antimony (V) Chloride

SbCl<sub>5</sub>

 $\begin{array}{c} {
m SbCl}_3 + {
m Cl}_2 = {
m SbCl}_5 \\ {
m 228.1} & {
m 70.9} & {
m 299.0} \end{array}$ 

A sample of SbCl<sub>3</sub> is melted in a ground joint Claisen flask. A gas inlet tube passes through one stopper. Chlorine is introduced, at first with heating and then without. When the liquid is saturated, the inlet tube is replaced by a capillary and the flask, which has a vacuum receiver, is evacuated. To remove the excess Cl<sub>2</sub>, a stream of air dried with  $P_2O_5$  is drawn through the apparatus by suction, followed by distillation of the flask contents at 14 mm. At this pressure, the SbCl<sub>5</sub> distills over at 68°C. The yield is 85%.

PROPERTIES:

Colorless liquid; fumes strongly in air. M.p.  $4.0^{\circ}$ C, b.p. (14 mm.)  $68^{\circ}$ C; (760 mm.) about  $140^{\circ}$ C (dec.);  $d_{20}2.346$ . On addition of a small quantity of H<sub>2</sub>O, forms the hydrates SbCl<sub>5</sub> · H<sub>2</sub>O and SbCl<sub>5</sub> . 4H<sub>2</sub>O. With addition of larger amounts of water, hydrolyzes to Sb<sub>2</sub>O<sub>5</sub>. Vapor pressure 6 mm. (51°C), 9 mm. (58°C).

**REFERENCES:** 

O. Ruff. Ber. dtsch. chem. Ges. <u>42</u>, 4026 (1909);
W. Biltz and K. Jeep. Z. anorg. allg. Chem. <u>162</u>, 34 (1927);
F. Seel. Z. anorg. Chem. <u>252</u>, 35 (1944).

## Antimony (III) Oxide Chloride

#### **SbO**Cl

 $SbCl_3 + H_2O = SbOCl + 2 HCl$ 228.1 18.0 173.2 72.9

A mixture of 70 ml. of  $H_2O$  and 100 g. of SbCl<sub>3</sub> is prepared, thoroughly stirred, and allowed to stand for several days. The crystals that precipitate are suction-filtered, pressed, washed with ether, and dried. Larger crystals are prepared, according to Edstrand, by heating SbCl<sub>3</sub> with absolute alcohol in a sealed tube at  $150^{\circ}C$ .

**PROPERTIES:** 

Colorless monoclinic crystals or crystalline powder. Soluble in hydrochloric and tartaric acids and in  $CS_2$ . Addition of water results in hydrolysis to  $Sb_2O_3$ . Heating to about  $250^{\circ}C$  results in formation of  $Sb_2O_5Cl_2$ . Above  $320^{\circ}C$  the product is  $Sb_2O_3$ .

**REFERENCES:** 

- A. Sabanjew. Z. f. Chem. 14, 206 (1871);
- J. M. van Bemmelen, P. A. Meerburg and U. Huber Noodt. Z. anorg. allg. Chem. 33, 290 (1903);
- E. Montignie. Bull. Soc. Chim. France Mém. (5) <u>14</u>, 378 (1947);
- M. Edstrand, Ark. Kemi <u>6</u>, 89 (1954).

## Hexachloroantimonic (V) Acid

#### HSbCl<sub>6</sub> · 4.5 H<sub>2</sub>O

 $\begin{array}{l} Sb_2O_3 + 8\,HCl + 2\,Cl_2 + 6\,H_2O = 2\,HSbCl_6\cdot 4.5\,H_2O \\ 291.5 \quad 291.7 \quad 141.8 \quad 108.1 \quad 833.2 \end{array}$ 

A sample of  $Sb_2O_3$  is dissolved in twice its weight of concentrated hydrochloric acid, and saturated with  $Cl_2$ . When the solution becomes greenish yellow, it is concentrated somewhat on a water

bath, gaseous HCl is added, and the mixture is allowed to crystallize overnight at 0°C in a desiccator over  $H_2SO_4$  (seeding should be used if possible). The crystals that precipitate are suctionfiltered on a glass suction filter and repeatedly washed with icecold concentrated hydrochloric acid. By concentration, additional crystal fractions can be obtained from the mother liquor and the wash liquids, after another treatment with HClgas. The crystalline fractions thus obtained can then be repeatedly recrystallized from concentrated hydrochloric acid (seed crystals should be set aside). The compound  $KSbCl_6 \cdot H_2O$  precipitates as greenish octahedral crystals if KCl, instead of HCl, is added to the concentrated solution.

## PROPERTIES:

Formula weight 416.6. Greenish, very hygroscopic prisms. M.p.  $44^{\circ}$ C. Effloresces over concentrated  $H_2SO_4$ . Readily soluble in ethanol, acetone, glacial acetic acid, and small amounts of  $H_2O$ . Hydrolyzes in excess  $H_2O$ . Alcohol and acetone solutions give a neutral reaction.

**REFERENCES:** 

- R. F. Weinland and H. Schmidt. Z. anorg. allg. Chem. <u>44</u>, 37 (1905);
- E. Groschuff. Z. anorg. allg. Chem. 103, 147 (1918);
- F. Seel. Z. anorg. Chem. 252, 24 (1943).

## Nitrosyl Chloroantimonate (V)

## NO(SbCl<sub>6</sub>)

$$\underset{65.5}{\text{NOCl}} + \underset{299.0}{\text{SbCl}_5} = \underset{364.5}{\text{NOSbCl}_6}$$

A solution of two moles of NOCl in a small quantity of dried  $CCl_4$  is cooled in an ice-salt bath. Moisture must be absent. A solution of one mole of  $SbCl_5$  in a small quantity of CCl is added dropwise. The yellow precipitate is suction-filtered, washed with some cold  $CCl_4$ , and vacuum dried over  $P_2O_5$ .

The corresponding nitryl chloroantimonate (V) can be prepared, according to Seel, by co-condensation of  $SbCl_5$ ,  $Cl_2$ , and  $NO_2Cl$ . It is a colorless salt.

PROPERTIES:

Yellow crystals, can be sublimated in a  $CO_2$  stream at  $150^{\circ}C$ . Melts in a closed tube at  $170^{\circ}C$ . Decomposes in moist air. **REFERENCES:** 

- H. Rheinboldt and R. Wasserfuhr. Ber. dtsch. chem. Ges. <u>60</u>, 732 (1927);
- F. Seel. Z. anorg. Chem. 252, 24 (1943);
- F. Seel, J. Nógrádi and R. Posse. Z. anorg. allg. Chem. <u>269</u>, 197 (1952).

## Antimony (III) Bromide SbBr<sub>3</sub>

 $Sb + 3 Br = SbBr_3$ 121.8 239.7 361.5

Antimony and bromine react with the appearance of a flame. The material can be prepared in exactly the same manner as described for As: a N<sub>2</sub> stream, saturated with Br<sub>2</sub>, is passed over Sb powder and the products are trapped in a receiver. Another method is to add very small portions of Sb powder through a vertical reflux condenser set on a Pyrex flask containing a solution of Br<sub>2</sub> in CS<sub>2</sub>, Then the Br<sub>2</sub> excess and the CS<sub>2</sub> are distilled off. The flask contents are distilled in a sausage distillation flask. Purification may be accomplished by recrystallization from CS<sub>2</sub> or, better still, via the Jander and Weis procedure, by redistillation over Sb powder and KBr in a stream of dry CO<sub>2</sub>. The apparatus illustrated in Fig. 205a is employed. Flask  $\alpha$  contains SbBr<sub>a</sub>, placed over Sb powder and KBr. A very large first cut is distilled into receiver v, while dry CO<sub>2</sub> is passed through the apparatus from b. Then v is removed, CO<sub>2</sub> is introduced in at the right, and the major fraction is distilled from b onto the chemically pure Sb. The material is again distilled from b, and again a large first cut is discarded. Finally, the main fraction is distilled into receivers attached with ground glass joints.

PROPERTIES:

White crystalline mass. When fused, it is bright, amber yellow. M.p. 96°C, b.p. 288°C; d 4.148. Hygroscopic. Hydrolyzes in  $H_2O$ . Rhombic.

**REFERENCES:** 

J. Niklés. J. Pharm. (3) <u>41</u>, 145 (1862);
A. C. Vournasos. Z. anorg. allg. Chem. <u>192</u>, 372 (1930);
W. Biltz and A. Sapper. Z. anorg. allg. Chem. <u>203</u>, 282 (1932);

E. Dönges. Z. anorg. allg. Chem. <u>263</u>, 112 (1950);
G. Jander and J. Weis. Z. Elektrochem. <u>61</u>, 1275 (1957).



Fig. 205a. Purification of antimony (III) bromide.

## Antimony (III) lodide SbI:

 $Sb + \frac{3}{2}I_2 = SbI_3$ 121.8 380.7 502.5

A solution of 14 g. of  $I_2$  in 300 ml. of toluene is refluxed with 7 g. of finely pulverized Sb until the iodine color disappears. The yellow-green solution is filtered from the unconverted Sb (best practice is to use an immersion suction filter and propel the fluid with compressed CO<sub>2</sub>) and allowed to crystallize, whereupon SbI<sub>3</sub> precipitates as red leaflets. These are freed of toluene in a vacuum desiccator at 40°C and then resublimed in a flow of CO<sub>2</sub> or in vacuum. The SbI<sub>3</sub> distills between 180 and 200°C. The yield is 80%.

**PROPERTIES:** 

Red, crystalline flakes; trigonal. In addition to the red trigonal form [d (22°C) 4.92], there are two greenish modifications, one rhombic and one monoclinic. d (17°C) 4.77. When fused, a pomegranate-red liquid. The vapor is orange red. M.p.  $170^{\circ}$ C, b.p.  $401^{\circ}$ C. Hydrolyzes in H<sub>2</sub>O to a yellow oxyiodide.

REFERENCES:

W. Biltz and A. Sapper. Z. anorg. allg. Chem. 203, 282 (1932);
J. C. Bailar and P. F. Cundy in: H. S. Booth, Inorg. Syntheses, Vol. I, p. 104, New York-London 1939.

## Ammonium Hexabromoantimonate (IV)

#### (NH<sub>4</sub>)<sub>2</sub>SbBr<sub>6</sub>

 $SbBr_3 + \frac{1}{2}Br_2 + 2NH_4Br = (NH_4)_2SbBr_6$ 361.5 79.9 195.9 637.3

A 4.9-g. sample of  $NH_{4}Br$  is dissolved in 50 ml. of concentrated (48%) hydrobromic acid with gentle warming and the solution is then cooled. A portion of the NH<sub>4</sub>Br precipitates as a very fine powder but redissolves on addition of 50 ml. of concentrated H<sub>2</sub>SO, (use cooling). A solution of 9 g. of SbBr<sub>3</sub> and 0.65 ml. of Br<sub>2</sub> in 10 ml. of concentrated hydrobromic acid is then added to the mixture. The (NH<sub>4</sub>)<sub>2</sub>SbBr<sub>5</sub> precipitates virtually quantitatively. It is suction filtered, washed three times with 4 ml. of concentrated hydrobromic acid, and dried in a vacuum desiccator over KOH. The drying process, usually complete after two days, involves loss of Br, and the substance takes on a dirty olive-green color. After drying, the composition of the substance corresponds approximately to the formula  $(NH_4)_2SbBr_{5,4}$ . If the drying is longer, the color becomes lighter and more bromine is lost. To rebrominate the substance, it is placed in a desiccator over concentrated  $H_2SO_4$  and kept there for a day under bromine vapor. The excess Br<sub>2</sub> is then evacuated. The preparation is then pure and corresponds to the formula (NH<sub>4</sub>)<sub>2</sub>SbBr<sub>6</sub>.

#### PROPERTIES:

Crystallizes in deep-black octahedra, stable in dry air. Moisture produces hydrolytic decomposition. Soluble in 2N hydrochloric acid; can be recrystallized (with loss of Br) from concentrated hydrobromic acid. Crystal structure: similar to  $K_2PtCl_{e}$ .

REFERENCES:

- F. Ephraim and S. Weinberg. Ber. dtsch. chem. Ges. <u>42</u>, 4450 (1909);
- K. A. Jensen. Z. anorg. allg. Chem. <u>232</u>, 193 (1937); <u>252</u>, 317 (1944);
- W. D. Schnell, Thesis, Freiburg, 1952.

## Antimony (III) Oxide

Sb<sub>2</sub>O<sub>3</sub>

 $\begin{array}{r} 2\,{\rm SbCl_3} + 3\,{\rm H_2O} = {\rm Sb_2O_3} + 6\,{\rm HCl} \\ 456.2 & 54.0 & 291.5 & 218.8 \end{array}$ 

A solution of  $SbCl_5$  in some highly concentrated hydrochloric acid is diluted with water. The resulting precipitate is repeatedly

washed and decanted with water and is then boiled repeatedly with dilute ammonia until the solution is free of halogen ions. It is then decanted repeatedly with water, washed on a filter, and dried. Other antimony compounds may be used in similar fashion to prepare  $Sb_2O_3$  by hydrolysis.

PROPERTIES:

White, crystalline powder; cubic (senarmontite) or rhombic (valentinite); transformation point  $570^{\circ}C \pm 10^{\circ}C$ . Hydrolysis yields the metastable valentinite. When treated with alkali, converts gradually to senarmontite.  $d_4^{25}$  5.19 (cubic), 5.79 (rhombic); b.p. 1425°C; m.p. 655°C. Sublimes in vacuum at 400°C. Slightly soluble in H<sub>2</sub>O. Heating gives a reversible yellow color.

#### **REFERENCES:**

L. Vanino. Handb. d. präp. Chem., 2nd ed., Stuttgart 1925, p. 229;

E. J. Roberts and F. Fenwick. J. Amer. Chem. Soc. <u>50</u>, 2133 (1928);
M. C. Bloom and M. J. Buerger. Z. Kristallogr. <u>96</u>, 367 (1937);
M. J. Buerger and S. B. Hendricks. Z. Kristallogr. <u>98</u>, 29 (1938);
A. Simon. Z. anorg. allg. Chem. <u>165</u>, 38 (1927).

## Antimony (V) Oxide

## Sb<sub>2</sub>O<sub>5</sub>

I.

 $2 \operatorname{HSbCl}_{6} + 5 \operatorname{H}_{2} O = \operatorname{Sb}_{2} O_{5} + 12 \operatorname{HCl}_{671.0}$ 90.1 323.5 437.6

The  $Sb_2O_5$  required for the preparation of chemically pure Sb is made by hydrolysis of hexachloroantimonic acid. The latter is dissolved in some double-distilled water, and the cold solution is diluted with more water. Then C. P. ammonia is added and the solution is heated on a water bath until the  $Sb_2O_5$  precipitate settles. Decantation follows, then repeated washing with chemically pure water and another decantation. Finally, the precipitate is suction filtered through a paper filter. The substance is dried in a dish by heating on a water bath.

II. 
$$2 \operatorname{SbCl}_5 + 5 \operatorname{H}_2 O = \operatorname{Sb}_2 O_5 + 10 \operatorname{HCl}_{598,1} 90.1 323.5 364.7$$

A mixture of  ${\rm SbCl}_5$  in 20 to 25 times its weight of cold water is prepared. After a few hours, the precipitate is filtered and washed with cold water. It is dried at 275°C to constant weight. III. Another method of preparation is the precipitation of a potassium antimoniate solution with nitric acid, followed by thorough washing and drying as above.

PROPERTIES:

Fine, pale yellow powder. Cubic. Slightly soluble in  $H_2O.d$  3.78. Loses oxygen on heating above 300°C and gradually converts to  $Sb_2O_4$ . New studies have shown that the compound does not correspond fully to  $Sb_2O_5$ , but that it is always somewhat hydrated. When heated for a long time, its composition corresponds to the formula  $Sb_2O_5$ . SbOOH. It seems questionable whether pure  $Sb_2O_5$  exists at all.

**REFERENCES:** 

- K. Dihlström and A. Westgren. Z. anorg. allg. Chem. <u>235</u>, 153 (1937); 239, 57 (1938);
- A. Simon and E. Thaler. Z. anorg. allg. Chem. 162, 260 (1927).

## Hydrated Antimony (V) Oxide

 $Sb_2O_5 \cdot (H_2O)_x$ 

$$\begin{array}{c} 2 \operatorname{SbCl}_5 + x \operatorname{H}_2 O = \operatorname{Sb}_2 O_5 \cdot (\operatorname{H}_2 O)_{x-5} + 10 \operatorname{HCl} \\ 598.1 & x \cdot 18 & 323.5 + 18 \cdot {}_{(x-5)} & 364.7 \end{array}$$

I. Antimony (V) chloride (100 ml.) is added dropwise and with vigorous stirring and constant cooling to 0°C to 7.5 liters of icecold saturated chlorine water. The liquid becomes cloudy. The copious precipitate is repeatedly washed and decanted with nineliter portions of water and then purified for 23 days in a rapid dialyzer. The material is filtered through a leaf filter and spread on a clay plate, where it is left to dry in the air. After a number of months, the preparation analyzes as  $Sb_2O_5 \cdot 4.58 H_2O_5$ .

II. If precipitation is performed at  $100^{\circ}$ C with only two liters of the precipitation liquid, followed by purification and drying as described in I, the resulting crystals are larger and have the composition Sb<sub>2</sub>O<sub>5</sub> • 4.40 H<sub>2</sub>O.

PROPERTIES:

Insoluble or very slightly soluble in moderately concentrated alkalis and acids. Opaque, white, highly adsorptive mass. Iso-thermal dehydration at room temperature produces one definitely identified hydrate,  $3 \text{ Sb}_2\text{O}_5 \cdot 5\text{H}_2\text{O}$ , which can be prepared by heating for many days in a sealed tube at  $300^{\circ}\text{C}$ .

**REFERENCES:** 

G. Jander. Koll. Z. 23, 130 (1918);
G. Jander and A. Simon. Z. anorg. allg. Chem. 127, 71 (1923);
A. Simon and E. Thaler. Z. anorg. allg. Chem. 161, 116 (1927).

## **Diantimony Tetroxide**

Sb<sub>2</sub>O<sub>4</sub>

 $2 \operatorname{Sb}_{2}O_{5} = 2 \operatorname{Sb}_{2}O_{4} + O_{2}$ 647.0 615.0

Antimony (V) oxide is ignited to red heat for about two weeks in a Pt crucible at  $800-900^{\circ}$ C. Other method: boiling Sb<sub>2</sub>O<sub>3</sub> with nitric acid, evaporating until fuming, and igniting as above.

**PROPERTIES:** 

Minute, lustrous crystals, yellow when heated. Infusible. Decomposes to  $Sb_2O_3$  and  $O_2$  at very high temperatures. Virtually insoluble in water, dilute acids and alkalis. Soluble in hot concentrated HCl and concentrated  $H_2SO_4$ . d. 6.6-7.5. Cubic or rhombic crystals.

**REFERENCES:** 

- K. Dihlström and A. Westgren. Z. anorg. allg. Chem. <u>235</u>, 153 (1937); <u>239</u>, 57 (1938).
- A. Simon and E. Thaler. Z. anorg. allg. Chem. 162, 260 (1927).

## Antimony(III) Sulfate

Sb<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub>

 $Sb_2O_3 + 3 H_2SO_4 = Sb_2(SO_4)_3 + 3 H_2O_{291.5} 294.2 531.7 54.0$ 

Antimony (III) oxide is dissolved in hot concentrated  $H_2SO_4$ . Long, silky needles of  $Sb_2(SO_4)_3$  precipitate on cooling. These are suction filtered on a glass filter crucible and dried on a clay dish. The substance can be obtained free of  $H_2SO_4$  by washing with xylene. Washing is continued until the wash liquor obtained by shaking the xylene with water no longer gives an acid reaction. The crystals are converted to an amorphous mass on washing, but the salt is then very pure. **PROPERTIES:** 

Colorless crystals, deliquesce in air. Cold water decomposes the substance to a basic sulfate. Complete hydrolysis results on boiling with water. d 3.62.

REFERENCE:

S. Metzl. Z. anorg. allg. Chem. 48, 143 (1906).

## Antimony (III) Oxide Sulfate

 $(SbO)_2SO_4$ 

 $\begin{array}{c} {\rm Sb}_2({\rm SO}_4)_3 + 2 \ {\rm H}_2{\rm O} = ({\rm SbO})_2{\rm SO}_4 + 2 \ {\rm H}_2{\rm SO}_4 \\ 531.7 \qquad 36.0 \qquad 371.6 \qquad 196.2 \end{array}$ 

Antimony sulfate is treated with 10 times its weight of cold water, the mixture is thoroughly stirred and allowed to stand overnight in the cold. It is then suction-filtered and dried at  $100^{\circ}$ C.

**PROPERTIES:** 

White powder, insoluble in water, soluble in dilute tartaric acid.

**REFERENCE:** 

S. Metzl. Z. anorg. allg. Chem. 48, 143 (1906).

## Sodium Thioantimonate (V)

 $Na_{3}SbS_{4} \cdot 9 H_{2}O$ 

 $\begin{array}{l} Sb_2S_3 + 8 \ NaOH + 6 \ S = 2 \ Na_3SbS_4 + Na_2SO_4 + 4 \ H_2O \\ 339.7 \quad 320.0 \quad 192.4 \quad 638.0 \quad 142.0 \quad 72.1 \end{array}$ 

1. A solution of 15 g. of SbCl<sub>3</sub> in 600 ml. of dilute hydrochloric acid is prepared. If a precipitate is produced as a consequence of hydrolysis, concentrated hydrochloric acid is added until the solution becomes clear. Then  $H_2S$  is bubbled through the solution, and the precipitate of  $Sb_2S_3$  is filtered off. It is mixed with 60 ml. of 20% NaOH; 6 g. of S (powder form) is added and the mixture is boiled with constant stirring until the orange-red color turns yellow. The water lost on boiling should be replaced from time to time. The solution is filtered through a fluted filter and evaporated until crystallization begins. If the solution becomes turbid, a few drops of 20% NaOH are added until it clears. After complete cooling, the crystalline precipitate is suction-filtered, washed with some alcohol, and dried in a desiccator over quicklime to which a few drops of ammonium sulfide solution have been added. The mother liquor can be further concentrated. The preparation can be purified by recrystallization from weakly alkaline solution (a few milliliters of sodium hydroxide are added to the water).

SYNONYM:

Schlippe's salt.

PROPERTIES:

Formula weight (Na<sub>3</sub>SbS<sub>4</sub>  $\cdot$  9H<sub>2</sub>O) 481.14. Bright yellow, large, tetrahedral crystals. Effloresces readily in air.

**REFERENCES:** 

- C. F. von Schlippe. Schweiggers Journ. f. Chem. and Physik <u>33</u>, 320 (1821);
- E. Riesenfeld. Anorganisch-chemisches Praktikum [Laboratory Manual for Inorganic Chemistry], Leipzig, 1930, p. 238;
- H. and W. Biltz. Übungsbeispiele aus der unorganischen Experimentalchemie [Practical Problems in Experimental Inorganic Chemistry], Leipzig, 1920, p. 133;
- F. Kirchhof. Z. anorg. allg. Chem. 112, 67 (1920).

## Bismuth

#### Bi

The chemically pure bismuth oxide starting material is prepared from highest purity commercial bismuth nitrate. This is dissolved at 18°C in one half its weight of 8% nitric acid. An equal weight of concentrated nitric acid is added, and the solution is cooled to  $0-10^{\circ}$ C while well stirred. The resulting crystalline slurry is suction-filtered on a coarse fritted glass filter and washed with some ice-cold, concentrated nitric acid. Further quantities of nitrate may be obtained by concentrating the mother liquor. The purification is repeated as many times as necessary; the nitrate product is decomposed to the basic nitrate by heating in a porcelain dish at  $110^{\circ}$ C and this is then converted to the oxide by igniting. The oxide is mixed with half its weight of pure KCN and reduced in a porcelain crucible. Reduction in a flow of H<sub>2</sub> is less convenient. Impurities in the Bi metal thus purified cannot be detected by chemical or spectroscopic means.

Bismuth metal can be purified to a considerable extent by slow crystallization in a large diameter Vycor test tube under paraffin oil. It is permitted to cool, while stirring with a perforated glass ladle, and the precipitated crystals are scooped out of the melt (it must be borne in mind that bismuth expands on solidification and may burst the test tube. A dish is therefore placed underneath). The crystals are melted. This treatment removes all impurities except for Sb, which forms mixed crystals with the bismuth.

#### **PROPERTIES:**

Atomic weight 209.00. Brittle, lustrous metal. May be pulverized. M.p. 271°C, b.p. 1560°C; d. 9.80. Insoluble in hydrochloric acid, soluble in nitric acid. Rhombohedral crystals.

#### **REFERENCES:**

O. Hönigschmid and L. Birckenbach. Z. Elektrochem. 26, 403 (1920);

- J. Löwe. Z. analyt. Chem. 22, 498 (1883);
- R. Schneider. J. prakt. Chem. 50, 461 (1894);
- F. Mylius. Z. anorg. allg. Chem. 96, 237 (1916);
- A. Classen. Ber. dtsch. chem. Ges. 23, 938 (1890).

## Bismuth (III) Chloride

BiCl<sub>3</sub>

 $\begin{array}{c} 2 \operatorname{Bi} + 3 \operatorname{Cl}_2 = 2 \operatorname{BiCl}_3 \\ 418.0 \quad 212.7 \quad 630.7 \end{array}$ 

A boat containing Bi is placed in a Vycor tube and heated in an electric furnace. A two-way stopcock permits introduction of either pure  $N_2$  or  $Cl_2$ . The air is displaced with  $N_2$  and the apparatus is dried by heating and passage of  $N_2$ . Then the chlorine stream is introduced and the temperature is raised until the reaction begins. The BiCl<sub>3</sub> sublimes into the part of the tube that is cooled by a water jacket or with wet filter paper. After about an hour the formation of BiCl<sub>3</sub> ceases. The Cl<sub>2</sub> is displaced with  $N_2$ , and the chloride is quickly removed from the tube (Hönigschmid describes a special emptying device).

Other preparative methods: solution of  $Bi_2O_3$  in hydrochloric acid, drying on a water bath, and distilling the residue in a stream of  $CO_2$ .

PROPERTIES:

Formula weight 315.37. Colorless crystals. B.p. 447°C, m.p. 233°C; d. 4.75. Soluble in alcohol and hydrochloric acid. Hydrolyzes to BiOCl in  $H_2O$ . Deliquesces in air.

**REFERENCES:** 

- O. Hönigschmid and L. Birckenbach. Z. Elektrochem. <u>26</u>, 403 (1920); Ber. dtsch. chem. Ges. 54, 1889 (1921);
- A. Voigt and W. Biltz. Z. anorg. allg. Chem. 133, 293 (1924).

## **Bismuth Dichloride**

## BiCl<sub>2</sub>

 $\begin{array}{c} {\bf Bi+Hg_2Cl_2=BiCl_2+2\,Hg} \\ {\bf 209.0} & {\bf 472.1} & {\bf 279.9} & {\bf 401.2} \end{array}$ 

An equimolar mixture of  $Hg_2Cl_2$  and powdered Bi is heated for two hours in a sealed tube at 250°C. After heating, the BiCl<sub>2</sub> is separated from the Hg and reheated in the tube.

**PROPERTIES:** 

Microscopically small, black crystals, insoluble in organic solvents. M.p. 163°C; d 4.86. At 300°C, it decomposes to Bi + BiCl<sub>3</sub>. When heated in air, produces  $Bi_2O_3$ , BiOCl and Cl<sub>2</sub>. With H<sub>2</sub>O decomposes to BiOCl, Bi and HCl. Yields BiICl<sub>2</sub> with I<sub>2</sub>.

**REFERENCES:** 

A. Schneider. Pogg. Ann. <u>96</u>, 136 (1855);
E. Montignie. Bull. Soc. Chim. France [5] <u>4</u>, 588 (1937).

## **Bismuth Oxide Chloride**

## BiOCl

 $BiCl_3 + H_2O = BiOCl + 2 HCl$ 315.4 18.0 260.5 72.9

A solution of 3 g. of  $Bi_2O_3$  in 300 ml. of hydrochloric acid (d 1.05) is heated to boiling, at which point 2.5 liters of boiling water is added. Boiling is continued until the initial precipitate

622

has redissolved. The solution is then allowed to cool until crystallization sets in. The BiOCl precipitate is filtered by suction.

PROPERTIES:

Colorless, crystalline powder, very slightly soluble in  $\rm H_2O_{\bullet}$  d 7.72. Tetragonal.

REFERENCES:

A. de Schulten. Bull. Soc. Chim. France [3] <u>23</u>, 156 (1900); W. Herz. Z. anorg. allg. Chem. <u>36</u>, 346 (1903).

#### Bismuth (III) Bromide

BiBr<sub>3</sub>

$$\begin{array}{c} 2 \text{ Bi} + 3 \text{ Br}_2 = 2 \text{ BiBr}_3 \\ 418.0 \quad 479.5 \quad 897.5 \end{array}$$

The same type of reaction vessel is employed as in the preparation of BiCl<sub>3</sub>. A stream of N<sub>2</sub> dried with concentrated H<sub>2</sub>SO<sub>4</sub> is passed through a wash bottle filled with Br<sub>2</sub>. This flask is slightly heated by a surrounding warm water bath. The N<sub>2</sub>-Br<sub>2</sub> mixture then passes through a P<sub>2</sub>O<sub>5</sub> drying tube. The apparatus used for preparation of this gaseous mixture is illustrated in Fig. 206.



Fig. 206. Generation of a stream of dry bromine vapor for preparation of bismuth (III) bromide.

Other method: Bismuth powder is placed in a retort,  $Br_2$  is poured over it, and the mixture is allowed to stand for a few days and then distilled. It is also possible to heat  $Br_2$  with Bi in a roundbottom flask equipped with a reflux condenser. In both cases purification is by vacuum distillation.

PROPERTIES:

Formula weight 448.75. Orange-yellow, crystalline mass. M.p. 218°C, b.p. 441°C; d. 5.7. Produces BiOBr with H<sub>2</sub>Q.
**REFERENCES:** 

- O. Hönigschmid and L. Birckenbach. Z. Elektrochem. <u>26</u>, 403 (1920);
- V. Meyer. Liebigs Ann. Chem. 264, 122 (1891).

## Bismuth Oxide Bromide BiOBr

 $\begin{array}{l} BiBr_{3} + H_{2}O = BiOBr + 2 HBr \\ 448.7 \quad 18.0 \quad 304.9 \quad 161.8 \end{array}$ 

A solution of 3 g of  $\text{Bi}_2O_3$  in 50 ml. of hydrobromic acid (d 1.38) is heated to boiling. It is then diluted with 1.5-1.6 liters of boiling water and boiling is continued until the initial precipitate redissolves. The BiOBr crystallizes on cooling in ice. It is washed with dilute hydrobromic acid and then with pure water.

PROPERTIES:

Colorless square crystals or colorless crystalline powder. Very stable, melts at red heat. Soluble in concentrated hydrobromic acid.

## **Bismuth (III) Iodide**

#### BiI₃

 $\begin{array}{rl} Bi + \frac{3}{2} I_2 = Bi I_3 \\ 209.0 & 380.7 & 589.7 \end{array}$ 

I.

lodine (20 g.) is ground with 45 g. of finely powdered Bi in a mortar and then rapidly placed in a retort. The mixture is heated until reaction begins. At the end of the reaction, a stream of  $CO_2$  is passed through the warm retort to remove unreacted iodine. The iodide is then sublimed in a stream of  $CO_2$ . Smaller quantities can also be prepared by placing the mixture in a boat inserted into a Vycor tube, heating, and then subliming the Bil<sub>3</sub> in a  $CO_2$  stream.

II.  $BiCl_s + 3 HI = BiI_s + 3 HCl_{315.4} - 383.7 - 589.7 - 109.4$ 

A solution of  $BiCl_3$  in hydrochloric acid is precipitated with concentrated hydriodic acid. The precipitate is filtered in a fritted glass suction funnel and washed free of Cl ions with concentrated hydriodic acid. The crystals are dried in vacuum over  $P_2O_5$ . They are then heated in vacuum nearly to the melting point and finally sublimed by stronger heating.

PROPERTIES:

Dark crystals with metallic luster. Very similar in appearance to iodine. M.p. slightly above 400°C. Slightly soluble in alcohol, benzene, and toluene.

REFERENCES:

L. Birckenbach. Ber. dtsch. chem. Ges. <u>40</u>, 1404 (1907); E. Dönges. Z. anorg. allg. Chem. 263, 112 (1950).

## **Bismuth Oxide Iodide**

BiOI

 $\begin{array}{c} \text{BiI}_{3} + 2 \text{ KOH} = \text{BiOI} + 2 \text{ KI} + \text{H}_{2}\text{O} \\ 589.7 \quad 112.2 \quad 351.9 \quad 332.0 \quad 18 \end{array}$ 

A precipitate of BiOI is formed when  $BiI_3$  is shaken with some dilute KOH. It is suction-filtered, washed and dried.

Other preparative methods: A solution of 20 g. of  $BI(NO_3)_3$  in 30 g. of glacial acetic acid is stirred into a cold solution of 7 g. of KI and 10 g. of sodium acetate in 400 ml. of  $H_2O$ . When the precipitate becomes brick red, it is filtered off, washed and dried.

A solution of 0.25 g. of  $\text{Bi}_2\text{O}_3$  in 40 ml. of hydriodic acid (d. 1.2) is diluted with six liters of water. The whole is heated on a water bath; crystalline BiOI immediately precipitates.

A solution of 30.5 g. of  $BiONO_3$  in 30 ml. of concentrated nitric acid is mixed with 60 ml. of  $H_2O$  and 60 ml. of glycerol. The solution is stirred with 125 ml. of 25% sodium hydroxide and 175 ml. of  $H_2O$ . The clear solution is diluted with  $H_2O$  to one liter. Then a solution of 33 g. of KI in 50 ml. of  $H_2O$  plus 60 ml. of acetic acid is added. After two hours it is suction filtered and washed. The yield is 32 g. of 91% pure material.

#### PROPERTIES:

Brick-red crystalline powder or copper-colored crystals. Only slightly attacked by hot water. Fuses at red heat with partial decomposition.

#### **REFERENCES:**

A. deSchulten. Bull. Soc. Chim. France [3] 157 (1900);

B. Fischer. Die neueren Arzneimittel [New Pharmaceuticals], 3rd ed., p. 20.

- F. François and M. L. Delwaulle. Bull. Soc. Chim. France [4] 53, 1104 (1933);
- Q. Minigoja and M. P. de Almeida. Arquiv Biol. (São Paulo) <u>26</u>, 182 (1942) [Chem. Abstr. 1944, 48 759].

## **Bismuth Oxide Nitrite**

#### **BiONO**<sub>2</sub>

 $\begin{array}{ccc} Bi(NO_3)_3 \cdot 5 \ H_2O \ + \ NaNO_2 \ = \ BiONO_2 \ + \ 2 \ HNO_3 \ + \ NaNO_3 \ + \ 4 \ H_2O \\ 485.1 & 69.0 & 271.0 & 126.0 & 85.0 & 72.1 \end{array}$ 

A mixture of 48.4 g. of crystalline bismuth (III) nitrate and 18.2 g. of mannitol is ground in a mortar. The mass is kneaded thoroughly until it is viscous and sticky. It is then allowed to stand until it can be readily stirred with the pestle. Then 100 to 300 ml. of water is added. (The mixture must not be allowed to stand for a long time without water, since it will decompose with generation of NO<sub>2</sub>.) The mixture is stirred until dissolved; it is filtered and a solution of NaNO<sub>2</sub> added. A thick slurry of  $BiONO_2 \cdot 1/2H_2O$ crystals precipitates. The slurry is suction-filtered and washed until it no longer shows an acid reaction. It is then dried on clay.

**PROPERTIES:** 

Formula weight (BiONO<sub>2</sub>  $\cdot$  1/2H<sub>2</sub>O) 280.0. Yellow-white crystals. Loses NO<sub>2</sub> above 60°. Loses its water of crystallization in vacuum over H<sub>2</sub>SO<sub>4</sub>.

**REFERENCE:** 

L. Vanino and E. Hartl. J. prakt. Chem. [2] 74, 150 (1906).

## Bismuth (III) Phosphate BiPO₄

 $\begin{array}{l} {\rm Bi(NO_3)_3} + {\rm H_3PO_4} = {\rm BiPO_4} + 3\,{\rm HNO_3} \\ {}_{(5\,{\rm H_2O})} \\ {}_{485.1} & 98.0 & 304.0 & 189.0 \end{array}$ 

I. A bismuth (III) nitrate solution (prepared as described above under BiONO<sub>2</sub>) is precipitated by  $H_3PO_4$  or a phosphate solution. The precipitate is removed by suction filtration and washed. II. A mixture of 15 g. of Bi(NO<sub>3</sub>)<sub>3</sub> · 5H<sub>2</sub>O and 7 g. of Na<sub>2</sub>HPO<sub>4</sub> ·

II. A mixture of 15 g. of Bi(NO<sub>3</sub>)<sub>3</sub>  $\cdot$  5H<sub>2</sub>O and 7 g. of Na<sub>2</sub>HPO<sub>4</sub>  $\cdot$  12H<sub>2</sub>O is placed in a large flask with some water and concentrated

nitric acid and heated on a water bath. Water is added dropwise until microscopic crystals precipitate.

PROPERTIES:

When prepared as described in method I, it contains three moles of water of crystallization. Preparations produced as described in method II contain no water of crystallization. Does not melt on heating. Only slightly soluble in water and dilute acids. Not hydrolyzed by boiling water. d. 6.323. The anhydrous form is monoclinic.

REFERENCE:

L. Vanino and E. Hartl. J. prakt. Chem. [2] 74, 151 (1906).

## **Bismuth (III) Borate**

 $BiBO_3 \cdot 2 H_2O$ 

4 Bi(NO <sub>3</sub> ) <sub>3</sub>	$+ Na_{2}B_{4}O_{7} =$	$= 4 \operatorname{BiBO}_3$	+ 10 HNO <sub>3</sub>	$+ 2 \text{NaNO}_3$
$(5 H_2 O)$	$(10 H_2 O)$	$(2 H_2 O)$		
1940.4	381.4	1215.4	630.2	170.0

A solution of bismuth (III) nitrate and mannitol, prepared as described above under  $BiONO_2$ , is precipitated with borax solution. The finely granular crystal powder is removed by suction, washed and dried.

PROPERTIES:

Formula weight (BiBO<sub>3</sub>  $\cdot$  2H<sub>2</sub>O) 303.85. White powder. Attacked by water and alkalis.

REFERENCE:

L. Vanino and E. Hartl. J. prakt. Chem. [2] 74, 151 (1906).

#### Sodium Bismuthate

#### NaBiO<sub>3</sub>

A) ANHYDROUS

$$\begin{array}{l} \text{Bi}_2\text{O}_3 + 3\text{ Na}_2\text{O}_2 = 2\text{ Na}\text{Bi}\text{O}_3 + \frac{1}{2}\text{ O}_2 + 2\text{ Na}_2\text{O}\\ 466.0 \quad 234.0 \quad 560.0 \quad 124.0 \end{array}$$

Dry  $Bi_2O_3$  is mixed with half its weight of  $Na_2O_2$  and heated in a vessel protected from access of moisture and  $CO_2$ . The initial heating (to  $350^{\circ}$ C) is rapid. It is then continued to about  $600^{\circ}$ C until gas generation ceases. The cooled sample should evolve no gas when carefully immersed in water. It is preferable to work with sintered magnesia crucibles.

PROPERTIES:

Yellow powder. Rapidly oxidizes acidified manganese (II) sulfate solution to permanganate.

#### **B) HYDROUS**

 $\begin{array}{rl} Bi_2O_3 \ + \ 6 \ NaOH \ + \ 2 \ Br_2 \ = \ 2 \ NaBiO_3 \ + \ 4 \ NaBr \ + \ 3 \ H_2O \\ 466.0 & 240.0 & 319.7 & 560.0 & 411.6 & 54.0 \end{array}$ 

A suspension of 170 g. of  $Bi_2O_3$  in 1.5 liters of 40% sodium hydroxide is vigorously stirred and oxidized at the boiling point by gradually added  $Br_2$  (300 g.). The brown precipitate formed is filtered off, washed with 40% sodium hydroxide, and suspended in three liters of  $H_2O$ . The suspension is now agitated for a while, until the color changes from brown through light brown to yellow. The precipitate is then allowed to settle; it is filtered, added to 1.5 liters of 53% NaOH, and refluxed for one half hour. The resultant brown precipitate is readily filtered off after settling. It is washed with 50% sodium hydroxide, placed, while still damp, in three liters of  $H_2O$ , and briefly agitated. When the yellow precipitate settles, it is filtered, thoroughly washed with water, and finally dried on clay. The yield is 170 g.

**PROPERTIES:** 

Formula weight 280.0. Fine yellow needles. Variable water content may go as high as  $5 H_2O$ . The usual formula is NaBiO<sub>3</sub> · 3.5 H<sub>2</sub>O. Reacts with acids with partial decomposition and formation of higher bismuth oxides; oxidizes Mn(II) in H<sub>2</sub>SO<sub>4</sub> to MnO<sub>4</sub> in the cold.

**REFERENCES:** 

E. Zintl and K. Scheiner. Z. anorg. allg. Chem. 245, 32 (1940);

R. Scholder and H. Stobbe. Z. anorg. allg. Chem. <u>247</u>, 392 (1941); H. Martin-Frere. Comptes Rendus Hebd. Seances Acad. Sci. 213,

436 (1941).

## Potassium Bismuthate

## KBiO<sub>3</sub>

 $\begin{array}{l} \text{Bi}_2\text{O}_3 \ + \ 6 \ \text{KOH} \ + \ 2 \ \text{Br}_2 \ = \ 2 \ \text{KBi}\text{O}_3 \ + \ 4 \ \text{KBr} \ + \ 3 \ \text{H}_2\text{O} \\ 466_{9}0 \ \ 336.6 \ \ 319.7 \ \ 592.2 \ \ 476.1 \ \ 54.0 \end{array}$ 

A suspension of 165 g. of  $Bi_2O_3$  in 1.5 liters of 50% potassium hydroxide is oxidized at the boiling point (as described above for

NaBiO<sub>3</sub>) with small portions of Br<sub>2</sub> (total 500 g.). A dark violet precipitate results. Now, an additional 500 ml. of hot 40% potassium hydroxide is added and the material is filtered after settling. The precipitate is washed with 40% potassium hydroxide, suspended in 3-5 liters of H<sub>2</sub>O, and agitated for a while. The liquid is decanted after some hours; the solid is washed with cold H<sub>2</sub>O and filtered. The bright red precipitate is dried over H<sub>2</sub>SO<sub>4</sub> in a desiccator. The yield is 205 g. of KBiO<sub>3</sub>  $\cdot$  1/3 H<sub>2</sub>O. Formula weight (an-hydrous) 296.1.

**REFERENCE:** 

R. Scholder and H. Stobbe. Z. anorg. allg. Chem. 247, 392 (1941).

## **Dibismuth Tetroxide**

#### Bi<sub>2</sub>O<sub>4</sub>

 $\begin{array}{c} 2 \text{ KBiO}_3 + 2 \text{ HClO}_4 = 2 \text{ KClO}_4 + \text{Bi}_2\text{O}_4 + \text{O}_2 \\ 592.2 \quad 200.9 \quad 277.1 \quad 482.0 \quad 32 \end{array}$ 

Potassium bismuthate (see above) (50 g.) is boiled for about 10 hours in a large excess of 10% perchloric acid, until a slight residue of only 1-2 g. of an orange-red precipitate is left. The precipitate is filtered off, washed and dried. This is hydrated Bi<sub>2</sub>O<sub>4</sub>.

**REFERENCE:** 

R. Scholder and H. Stobbe. Z. anorg. allg. Chem. 247, 392 (1941).

## SECTION 11

# a) Elemental Carbon

## U. HOFMANN

Only processes of special importance to the laboratory will be discussed below. The corresponding industrial processes are not described.

The manufacture of diamonds [F. P. Bundy, H. T. Hall, H. M. Strong and R. H. Wentorf. Nature (London) <u>176</u>, 51 (1955)], which involves a great deal of expensive apparatus, will not be treated.

## Pure Carbon

The carbon available for industrial or laboratory purposes is not pure. It contains carbon compounds (with H, O, N, S), ashforming constituents and adsorbed gases or vapors. Preparation of highest-purity carbon, as well as its use in the pure state, may be achieved only in a high-vacuum apparatus [A. Stock et al. Z. anorg. allg. Chem. <u>195</u>, 158 (1931)]. Purity in excess of 99% is usually sufficient. This may be achieved by the following processes, which apply equally to coke-type carbon, carbon black, activated charcoal or graphite.

Removal of  $ash-forming\ mineral\ matter$ . Exhaustive boiling with dilute HNO<sub>3</sub> or dilute HCl; ignition to red heat at 900 to 1000°C in a stream of Cl<sub>2</sub>; treatment with hydrofluoric acid to remove silicates; calcining for several hours at 2000 to 3000°C in vacuum, in CO or in inert gas (this treatment results in structural changes, i.e., graphitization).

Removal of carbon compounds (coking). Solvents are incapable of removing all of these constituents. Complete removal is achieved by calcining for many hours in vacuum or in  $H_2$ ,  $N_2$  or inert gas at 900 to 1000°C. If more than a few percent of carbon compounds are present before calcination, the additional formation of C from these carbon impurities will result in structural changes. These structural changes tend chiefly to reduce adsorptive power. This difficulty can be overcome by activation (q.v.) without affecting the purity of the carbon.

Surface oxides (q.v.) can be completely removed as CO and CO<sub>2</sub> at 1000°C.

Removal of adsorbed gases and vapors. It must be borne in mind that adsorptive carbon can take up a few percent by weight of CO<sub>2</sub>, H<sub>2</sub>O, etc., at room temperature. Purification is achieved by heating for many hours at 300°C in high vacuum.

#### REFERENCES:

O. Ruff et al. Z. anorg. allg. Chem. 148, 313 (1925).

U. Hofmann et al. Z. anorg. allg. Chem. <u>255</u>, 195 (1947). U. Hofmann and G. Ohlerich. Angew. Chem. <u>62</u>, 16 (1950).

## **Special Carbon Preparations**

#### I. CARBON MONOXIDE BLACK

This carbon black deserves more attention because mediumsized graphite crystals of it undergo little aggregation. It is prepared by passing CO over finely divided, pure Fe (e.g., pure iron prepared by reduction of iron carbonyl) at 400-700°C. Depending on the temperature of preparation, the following average crystal sizes are obtained: 50 Å at 400°C, 100 Å at 500°C, 200 Å at 700°C. It should be remembered that any occluded Fe compounds can be removed only by a long purification process (see Pure Carbon).

## II. GRAPHITIZED CARBON BLACK

Calcination in vacuum or protective gas for many hours at 2000-3000°C causes the crystallites of the individual carbon black particles to grow into slightly larger crystals, without significantly affecting the size of the particles themselves (channel black about 200 Å, lamp black 500-2000 Å, Thermatomic black about 3000 Å). The product of the calcination is a polyhedron consisting of slightly pyramidal graphite crystals with their vertices directed inward, with its surface composed of the basal planes (001) of graphite.

## III. GRAPHITE OXIDE BLACK

This black consists of very thin graphite foils (about 20 to 50 Å thick) the diameter of which, however, may range up to some hundredths of a millimeter, depending on the starting graphite. It is prepared by making graphite oxide from graphite (see Graphite Oxide). The graphite oxide is decomposed by rapid heating to  $300-400^{\circ}$ C, where it deflagrates. Since the oxygen bound in the graphite oxide evolves as CO and CO<sub>2</sub> in the deflagration, it is possible that the hexagonal network of the carbon black foils contains holes of atomic dimensions.

A better product of otherwise identical properties can be prepared as follows: An alkaline suspension of graphite oxide is reduced with hydroxylamine at  $80^{\circ}$ C. The agglutination of the foils upon filtering and desiccation can be substantially inhibited by freeze-drying of the salt-free, dialyzed suspension.

In both cases the carbon black contains only about 80% carbon. It also contains O, H, etc., and must be carefully purified by coking and, if necessary, also freed of mineral matter.

#### IV. LUSTROUS CARBON

Brittle foils with perfectly reflecting surfaces because the basal planes of the graphite crystals are parallel to the foil planes. Surfaces up to several centimeters. Thicknesses up to some tenths of a millimeter. Coatings of lustrous carbon on ceramic materials serve as high electrical resistances. The crystal size is about 25 Å.

The material is prepared by cracking dilute hydrocarbons (e.g., propane vapor at about 10 mm. or  $N_2$  saturated with gasoline vapor at room temperature) at 800 to  $1000^{\circ}$ C. The material is deposited on a smooth surface such as porcelain or quartz. It is desirable to add some  $O_2$  or water vapor to the hydrocarbon, so as to avoid simultaneous precipitation of reactive, rough-surface carbon. After a thickness of some hundredths of a millimeter has been achieved, the carbon foils either flake off by themselves or may be readily detached after cooling. The best tightly adhering coatings are produced on surfaces that are not completely smooth.

## V. GRAPHITE FOILS AND FILMS

a) A sol prepared from 1-2 g. of graphite oxide (q.v.) in 100 ml. of water (well shaken or stirred) is allowed to evaporate. The graphite oxide separates as a foil on the bottom of the vessel. Very careful, slow heating with gradually rising temperature causes the sheet to give off CO,  $CO_2$  and  $H_2O$  without deflagration until, at 1000°C, a graphitelike film is obtained. This still contains a few percent of O and H. The size and thickness of the foil is the same as that of the graphite oxide and can therefore be controlled by the amount and concentration of the graphite oxide sol and the size of the vaporization vessel. Films with surfaces as large as 50 cm<sup>2</sup> can be obtained in this manner.

b) Very pure graphite foils can be produced by igniting foils of lustrous carbon for many hours in vacuum or in CO. The

temperature is  $2000-3000^{\circ}$ C. The foils are of the same size as those of lustrous carbon, i.e., about 1 cm. in surface diameter.

## VI. ADSORPTIVE CARBON (ACTIVATION)

To achieve the best adsorptive power, it is helpful to break down the crystalline aggregations of carbon by careful oxidation. A carbon is considered highly adsorptive if, for example, it adsorbs its own weight of  $CCl_4$  at room temperature from half-saturated  $CCl_4$  vapor.

This breaking down of structure is effected most simply by calcining at  $950^{\circ}$ C in a stream of CO<sub>2</sub> or H<sub>2</sub>O vapor and is continued until half the carbon has burned off. It suffices to pass the CO<sub>2</sub> slowly over the carbon spread in a thin layer in a boat. Subsequent removal of the adsorbed CO<sub>2</sub> or H<sub>2</sub>O by heating for many hours at  $300^{\circ}$ C in high vacuum is recommended. If surface oxides have formed due to adsorbed air, they may be removed by heating to red heat.

**REFERENCES:** 

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- III. G. Ruess and F. Vogt. Mh. Chem. 78, 222 (1948).
- IV. K. A. Hofmann and U. Hofmann. Ber. dtsch. chem. Ges. 59, 2433 (1926); G. Ruess. Z. anorg. allg. Chem. 255, 263 (1947).
- Va. H. Thiele, Forschungen und Fortschritte 10, 408 (1934); German patent 600 768.
  - b. U. Hofmann. German patent 752 734.
- VI. O. Ruff and G. Rössner. Ber. dtsch. chem. Ges. <u>60</u>, 411 (1927);
   U. Hofmann et al. Z. anorg allg. Chem. <u>255</u>, 195 (1947).

## Surface Compounds of Carbon

This section is devoted to chemical compounds formed on the surface of the graphite crystal.\* The internal structure of the crystal is not significantly changed by the surface compounds, which is not the case with graphite compounds.

<sup>\*</sup>Chlorine is probably bound not at the surface of the graphite crystal but by hydrocarbon impurities present in the material (see DEGUSSA, Nachrichten aus Chemie und Technik, August 21, 1954).

I. OXYGEN COMPOUNDS REACTING AS ACIDS IN AQUEOUS SOLUTIONS

Preparation involves heating the carbon for many hours in a stream of  $O_2$  at a temperature as near as possible to the ignition point, i.e., 500°C. During oxidation, the carbon must be vigorously shaken or, better still, fluidized by the  $O_2$  stream. Since oxidation produces  $CO_2$ , which is then absorbed, it is recommended that the oxidation product be subjected to long heating at 300°C in high vacuum. It should be remembered that the oxidation is accompanied by activation with  $O_2$ .

If a highly adsorptive carbon is used, as much as 15 g, of  $O_2$  can be taken up by 100 g, of the carbon. Under the same conditions, steam produces acid groups whose concentration may reach 700 meq. of H<sup>+</sup> ions per 100 g, of the preparation. The material is tested by shaking 0.1 g, of the carbon with 100 ml, of 0.05N alcoholic KOH. The H<sup>+</sup> ions can be replaced by CH<sub>3</sub> groups through methylation with diazomethane. Because of the acidic surface oxides, the carbon is readily wetted by water and poorly by benzene, as contrasted with carbon having no acid surface oxides. Above 500°C,  $O_2$  is released as CO and CO<sub>2</sub>.

## II. OXYGEN COMPOUNDS REACTING AS BASES

IN AQUEOUS SOLUTIONS

These compounds are always formed when carbon comes into contact with air or  $O_2$  at room temperature. Their formation can only be avoided when contact is prevented. These basic compounds may coexist on the surface of the carbon with the acid-forming O compounds. With highly adsorptive carbon these compounds may exert, in aqueous solution, an effect equivalent to a concentration of 100 meq. of OH<sup>-</sup> ions per 100 g. of carbon.

The material is tested by shaking with 0.05 N HCl. The basic surface oxides probably participate in the catalytic decomposition of  $H_2O_2$  on carbon and when carbon is used as an oxygen electrode.

Above  $500^{\circ}$ C, the O is released as CO and CO<sub>2</sub>.

## III. SULFUR COMPOUNDS

Carbon and excess S are heated for two days at  $600^{\circ}$ C in a sealed tube. The product is then washed thoroughly in a Soxhlet apparatus with CS<sub>2</sub>, toluene and alcohol. A highly adsorptive carbon can take up as much as 30 g. of S per 100 g. of preparation. Above  $500^{\circ}$ C, the preparations release S, and as the temperature rises, CS<sub>2</sub> is also generated.

#### REFERENCES:

I. H. Kruyt and G. de Kadt. Kolloid-Z. 47, 44 (1929); U. Hofmann and G. Ohlerich. Angew. Chem. 62, 16 (1950). Regarding preparation with concentrated nitric acid, see U. Hofmann and G. Ohlerich, above.

- II. A. Frumkin. Kolloid-Z. <u>51</u>, 123 (1930); G. Brinkmann. Angew. Chem. <u>61</u>, 378 (1949).
- III. J. P. Wibaut and E. J. v.d. Kam. Rec. Trav. Chim. <u>49</u>, 121 (1930); R. Juza and W. Blanke. Z. anorg. allg. Chem. <u>210</u>, 81 (1933); U. Hofmann and G. Ohlerich. Angew. Chem. <u>62</u>, 16 (1950).

b) Graphite Compounds

W. RÜDORFF

## Alkali Graphite Compounds

These may be prepared with melts or vapor of the alkali metals K, Rb and Cs. Depending on the proportions and/or the reaction temperature, the compounds resulting have the following approximate compositions:  $C_{8}M$  (1st stage),  $C_{24}M$  (2nd stage),  $C_{36}M$  (3rd stage) and  $C_{48}M$  (4th stage).

 $C_8K$ 

The apparatus is that of Fig. 207. Section B of the approximately 22 mm. I.D. glass tube is charged with about 1 g. of powdered or ground graphite which prior to use has been thoroughly heated at 900°C in high vacuum. An ampoule or glass tube containing pure K metal is introduced into A in a stream of  $N_2$ , the open end facing a. After the reactor tube is sealed off at a, the K is distilled into B in high vacuum. Then an electrical heater maintained at 300°C is placed over A and B. It is desirable to rotate the tube in ground joint E after some time so that even the graphite particles adhering to the glass at the bottom may react as completely as possible. The conversion is complete when the preparation acquires a dark copper color and is homogeneous. If some blue or black particles have formed, then the heating period was too long or the temperature too high. If this is the case, some K is distilled back from C to B and the heating operation is repeated. After all the excess K has been distilled away from B, the material is permitted to cool. The apparatus is then filled with N<sub>2</sub> and sealed off at b and c and the tube is raised to a vertical position, so that the product is separated from the K-wetted glass wall C and collects at D. Here the attached storage tubes (only one of which is illustrated) are filled and then torch-sealed.

 $C_{24}K$ 

The preparation corresponds to that for  $C_8K$  but the furnace is kept at 360°C. The end of the conversion is identified by the



Fig. 207. Preparation of alkali graphite compounds.

uniformly steel-blue color of the preparation. Precautions must be taken to ensure that part of the preparation does not again turn brown by absorbing K vapor when the furnace is cooled. If this is the case, heating must be continued. If heating is continued for too long, the products are poor in K.

## C<sub>36</sub>M AND C<sub>48</sub>M

These are prepared from a stoichiometric mixture of graphite powder or flakes and K metal by heating for 20 hours at 300 to  $400^{\circ}$ C in an evacuated, sealed tube.

Rubidium graphite and cesium graphite are prepared in similar fashion.

Analysis. After weighing, the tubes containing the substance are cut open and the contents are discharged into a preheated Erlenmeyer flask ( $C_8M$  and  $C_{24}M$  either ignite or smolder in the flask). After cooling, the graphite is boiled with water, filtered and washed. The alkali in the filtrate is determined by volumetric analysis. The dried graphite, which still contains a few percent of alkali, is repeatedly boiled (until fuming) with concentrated  $H_2SO_4$  and then calcined. The ash content of the starting graphite is calculated from the weight of the alkali sulfate obtained.

#### PROPERTIES:

 $C_8K$  is dark copper-red. Larger crystals have a metallic bronze color.  $C_{24}K$  is steel blue with a metallic luster. The compounds poorer in alkali are blue-black to black. These materials are very sensitive to  $O_2$  and moisture. They ignite in air.

Structure of CsM: there is an expanded graphite lattice, in the c direction, with alkali layers in each vacancy of the C lattice. In  $C_{24}M$ ,  $C_{36}M$ ,  $C_{46}M$ , each second, third, or fourth layer vacancy of the C lattice is expanded by an intercalated alkali layer.

#### **REFERENCES:**

K. Fredenhagen and G. Cadenbach. Z. anorg. allg. Chem. <u>158</u>, 249 (1926).

K. Fredenhagen and K. Suck. Z. anorg. allg. Chem. <u>178</u>, 353 (1929).
A. Schleede and Wellmann. Z. phys. Chem. (B) <u>18</u>, 1 (1932).
W. Rüdorff and E. Schulze. Z. anorg. allg. Chem. <u>277</u>, 156 (1954).

## Alkali Ammine Graphite Compounds

Prepared from graphite and alkali metal dissolved in liquid ammonia.

About 2 g. of powdered or flake graphite and the stirring bar of a magnetic stirrer are introduced into vessel c, which has a fritted glass disc (Fig. 208). After careful drying of the apparatus

Fig. 208. Preparation of alkali ammine graphite compounds.

and displacement of the air with pure  $N_2$ ,  $NH_3$  (liquefied over Na in vessel  $\alpha$ ) is condensed in b on the alkali metal (Li, Na, K, Rb or Cs-at least 1/30 gram-atom). Stopcock  $h_2$  is closed and the solution is siphoned from b into c by cooling vessel c. To complete the reaction in c the contents are well stirred after removal of the cooling bath. This may cause the solution to boil (if this occurs, stopcock  $h_2$  is opened). The solution must still be blue. If it is not, additional alkali metal must be added. Finally, stopcock  $h_3$  is opened and the solution forced with  $NH_3$  pressure through the frit and into another, well-cooled vessel. To wash the graphite,  $NH_3$  is condensed in c once or twice and forced out of the vessel in the same manner as before. Then c is brought to room temperature. When no further  $NH_3$  escapes through the pressure release valve, the apparatus is purged with  $N_2$  and the



product is poured into tubes d, e and f, which are then sealed. Analysis: The contents of a weighed ampoule are placed in an Erlenmeyer flask containing 50-70 ml. of ice-cold 2N H<sub>2</sub>SO<sub>4</sub>. The decomposed product is boiled for one half hour and the graphite is filtered off and washed. The alkali is determined as sulfate in an aliquot of the filtrate; KOH is added to another aliquot, and the NH<sub>3</sub> is distilled off and determined by volumetric analysis. The dried graphite is treated as described for potassium graphite compounds.

PROPERTIES:

The compound approximates  $C_{12}M(NH_3)_2$ . Very dark blue, enlarged crystals. Very sensitive to air and moisture. The K, Rb and Cs compounds can be stored indefinitely, but the Li and Na compounds decompose in a few days to amides, evolving  $H_2$ .

REFERENCE:

W. Rüdorff, E. Schulze and O. Rubisch. Z. anorg. allg. Chem. <u>282</u>, 232 (1955).

## **Graphite Oxide**

Prepared by oxidation of graphite with  $KClO_3$  in a mixture of concentrated sulfuric and nitric acids.

Graphite (10 g.), as ash-free and uniform as possible, is suspended in a mixture of 175 ml. of concentrated  $H_2SO_4$  and 90 ml. of 68% HNO<sub>3</sub>. A total of 110 g. of KClO<sub>3</sub> is added in small portions over a period of several days to the cooled flask. The product is repeatedly stirred with 6 to 10 liters of distilled water, then settled and decanted until the wash water is neutral to litmus. As washing progresses the blue-green particles become brown and swell increasingly as the salts are removed. At the end of the washing the settling is slow and incomplete. (Purification requires several weeks, and is shortened somewhat by use of a centrifuge.) The brown slurry is suction-filtered as much as possible, spread out to dry in air, ground and dried to constant weight in vacuum at  $50^{\circ}C$  over  $P_2O_5$ .

A light, almost white graphite oxide is obtained by washing in the dark with 5% HCl, containing  $ClO_2$ . This product contains only about 0.5% ash, but after vacuum-drying still shows a very small amount of chlorine.

Good yields of products with higher oxide contents can be achieved only when a well-crystallized graphite is used as the starting material, since otherwise the resulting oxidation products

638

are mostly soluble. It is advisable to use a graphite of uniform particle size, in the range 0.01-0.3 mm. The coarser the particles, the longer the time required for oxidation.

Washing of the preparation with acetic acid and ether is not recommended, as this results not only in adsorption but in acetylation of the OH groups of the graphite oxides (see References, G. Ruess).

Additional preparative methods: A faster method (Hummers and Offeman) proceeds as follows: 100 g. of finely milled graphite and 50 g. of NaNO<sub>3</sub> are stirred together in 2.3 liters of ice-cooled concentrated  $H_2SO_4$ . With vigorous stirring, 300 g. of KMnO<sub>4</sub> is added over a period of many hours. The rate of addition is governed by the particle size of the graphite. The temperature of the mixture must not exceed 20°C. The mixture is then heated to 35°C and after 30 minutes is slowly reacted with 4.6 liters of water. The reaction temperature must not exceed 70°C. The hot suspension is then held at 70°C for another 15 minutes. After dilution to 14 liters, the MnO<sub>2</sub> and excess KMnO<sub>4</sub> are reduced with 3%  $H_2O_2$ . The yellow-brown suspension is filtered while still warm and washed as above.

Anodic oxidation in concentrated  $HNO_3$  results in graphite with only a low degree of oxidation.

For methylation of graphite oxide with diazomethane in ether or dioxane solution, see references (Hofmann and Holst; Ruess).

SYNONYMS:

Graphite oxyhydroxide, graphitic acid.

PROPERTIES:

No definite formula. The C:O atomic ratio varies between 6:1 and 6:2.5. The ratio between C atoms and OH groups is about 4:1.

The dry material is brown to black. Very hygroscopic because of intercrystalline swelling. With careful heating to  $100^{\circ}$ C and above, CO and CO<sub>2</sub> are split off. Deflagrates on rapid heating. Hydrogen ion content (graphitic acid) available for exchange is about 600 meq./100 g. Readily reduced to graphitelike products by Sn<sup>2+</sup>, Fe<sup>2+</sup>, HI, N<sub>2</sub>H<sub>4</sub>, NH<sub>2</sub>OH, etc. Resistivity of the dry preparation is, depending upon the O content, between 10<sup>3</sup> and 10<sup>7</sup> ohm  $\cdot$  cm.

Structure: Tetrahedrally inclined C planes with metastable bond at the O and simultaneous bonding of the OH groups.

Applications: Depolarizer in dry cells, membranes for measurement of the partial pressure of water vapor, membrane electrodes. REFERENCES:

- U. Hofmann and E. König. Z. anorg. allg. Chem. 234, 311 (1937).
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- A. Clauss, U. Hofmann and A. Weiss. Z. Elektrochem. <u>61</u>, 1284 (1957).
- W. S. Hummers and R. E. Offeman. J. Amer. Chem. Soc. <u>80</u>, 1339 (1958).

## Carbon Monofluoride

Prepared by fluorination of graphite at 400 to 500°C.

The fluorination apparatus consists of a Cu trap, cooled with liquid nitrogen to remove impurities in the  $F_2$  stream, and a long Cu reactor tube, about 2 cm. in diameter, screwed onto the trap. A cap, opening into a thin Cu tube 1 to 2 mm. in diameter, is screwed to the other end of the Cu tube.

About 1 g. of graphite is spread out in a thin layer in a Cu boat placed in the middle of the reactor tube. The thermocouple is insulated from the tube wall by a thin layer of asbestos. A thicker asbestos layer protects it from contact with the furnace windings. The hot junction is, of course, exposed.

The graphite is thoroughly heated at  $800 \text{ to } 900^{\circ}\text{C}$  in high vacuum for an hour or two immediately before the fluorination. Only after complete cooling in high vacuum is N<sub>2</sub> is admitted.

A slow  $F_2$  stream (3-6 liters/hr) is passed through the apparatus and over the graphite at room temperature for 20 minutes before the start of the fluorination proper. Only then is the furnace turned on. The reaction starts between 420 and 500°C, but the initial conversion proceeds quietly within a narrow range of only about  $30^\circ$ . The temperature at the onset of the reaction is significantly dependent upon the size of the graphite crystals and the HF and  $O_2$  content of the  $F_2$  stream. With very finely powdered graphite of <0.01 mm. particle size and with impure  $F_2$ , combustion to volatile carbon fluorides may occur even at 400°C. Preparations that are already well fluorinated will tolerate temperatures as high as  $520^\circ$ C toward the end of the reaction. Deflagration always occurs above  $550^\circ$ C.

The progress of the reaction is followed by cooling the graphite in a stream of  $F_2$  and weighing. If no weight gain occurs within an hour, the reaction may be presumed to be over. The total fluorination time is four to seven hours.

The yield, based on the graphite, is almost quantitative, but when fluorination is continued for longer periods, slight losses, due to combustion to volatile carbon fluorides, are sustained. The fluorine content computed from the weight gain is usually 1% less than the real content.

Analytical determination of F is accomplished by decomposition with metallic Na in a sealed iron tube at  $900^{\circ}$ C. The Na is removed with alcohol, and the F precipitated as PbFCl after filtering off the C.

SYNONYM:

Graphite fluoride.

PROPERTIES:

Depending on experimental conditions, the fluorination products have compositions varying between  $CF_{0.68}$  and  $CF_{0.99}$ . When the F content is low, the preparation is gray black, while at very high F contents it is silver white. Depending upon F content, the densities vary between 2.78 and 2.50 (in xylene). The resistivity is greater than  $10^{6}$  ohm  $\cdot$  cm.

The preparations are not wetted by water and are neutral to acids and bases. Hydrogen has no effect below  $400^{\circ}$ C. Reduced to graphite on boiling with Zn dust and glacial acetic acid. Above  $400^{\circ}$ C thermal decomposition to volatile carbon fluorides occurs. When heating is rapid and the temperatures high, this is accompanied by deflagration and flame.

Structure: Tetrahedrally inclined C planes with F atoms bonded above and below the C planes.

REFERENCES:

- O. Ruff. O. Brettschneider and E. Ebert. Z. anorg. allg. Chem. <u>217</u>, 1 (1934).
- W. Rüdorff and G. Rüdorff. Z. anorg. allg. allg. Chem. 253, 281 (1947).

## Tetracarbon Monofluoride

Prepared by fluorination of graphite at room temperature in the presence of HF.

A stream of  $F_2$  at 4-5 liters/hr. is passed through a copper wash flask held at 0°C and filled with anhydrous HF, and then through an attached Cu tube (about 2 cm. in diameter). A Cu boat containing a thin layer of about 1 g. of graphite is placed in the tube. The end of the tube is closed with a screw-on cap having a narrow opening. The reaction is finished after one or two hours, as shown by the constant weight of the product. To remove excess HF, the preparation is washed with dilute base, water and alcohol and dried to room temperature over soda-lime. The yield is quantitative.

Quantitative determination of F is carried out in the same manner as with carbon monofluoride.

**PROPERTIES:** 

The composition lies within the range  $C_{3,e}F$  to  $C_4F$ . The color of the preparation is velvety black, sometimes somewhat bluish. d (under xylene) 2.05-2.09. Resistivity 2-4 ohm  $\cdot$  cm. at 750 kg/cm<sup>2</sup>

Stable to acids, bases and the common organic reagents. Slowly decomposes when heated for a long period above  $100^{\circ}$ C. Deflagrates when rapidly heated over a flame, forming sootlike flakes.

Structure: Flat C layers as in graphite, with F atoms bonded above and below the C layers.

**REFERENCE:** 

W. Rüdorff and G. Rüdorff. Ber. dtsch. chem. Ges. 80, 417 (1947).

## **Graphite Salts**

Prepared by oxidation of graphite in the presence of anhydrous acids.

#### GRAPHITE BISULFATE

An agitated suspension of about 1 g. of graphite (coarse or finely crystalline) in 10 ml. of concentrated  $H_2SO_4$  is mixed with a solution of  $CrO_3$  or  $K_2Cr_2O_7$  in concentrated  $H_2SO_4$ . At least 3 meq. of active oxygen (100 mg. of  $CrO_3$  per g. of C) is required for complete oxidation of the graphite. After 15 minutes the graphite is uniformly dark blue and can be suction-filtered through a glass frit and washed with  $H_2SO_4$ . Removal of the adhering  $H_2SO_4$  may be accomplished only by very long washing with sirupy pyrophosphoric acid or quick washing with ice-cold dimethyl sulfate. The reaction is virtually quantitative when coarse crystalline graphite is used, but when fine crystalline carbon is employed, brown, colloidal oxidation products also appear.

When oxidation is incomplete or the blue bisulfate is reduced with Fe (II), Sn (II) or with graphite itself, the product is less highly oxidized and the color is no longer blue.

The analytical composition may be determined from the quantity of oxidants consumed in preparation, from the oxidizing effect upon reducing agents, or by determination of the sulfate content after washing with  $H_4 P_2 O_7$  or  $(CH_3)_2 SO_4$ .

Additional preparative methods: The graphite may also be oxidized with  $S_2O_8^{2-}$ , PbO<sub>2</sub>, HIO<sub>3</sub>, HIO<sub>4</sub>, Mn(III) and Mn(IV) compounds, as well as anodically.

#### PROPERTIES:

The composition of the blue compound corresponds (in acid) to about  $C_{24}^{+}HSO_{4}^{-} \cdot 2H_{2}SO_{4}$ .

The product is as crystalline as the initial graphite, but swollen and very dark blue. It may be stored only under concentrated acid. Decomposed immediately by water, humid air, alcohol, ether, acetone and benzene.

STRUCTURE:

In the blue bisulfate, layers of acid anions and molecules are intercalated between the C planes. In the less oxidized black products, acid layers are intercalated in regular sequence in each second, third, fourth, etc., layer vacancy.

## GRAPHITE NITRATE, GRAPHITE PERCHLORATE

Prepared by washing graphite bisulfate (see above) with fuming nitric acid (d 1.52) or with 70%  $HClO_{4_2}$  respectively, until the filtrate is free of sulfate ions, or by oxidation of graphite in the corresponding acids,  $N_2O_5$  being present in the case of nitrate and  $CrO_3$  in the case of perchlorate. The conversion of the bisulfate is reversible.

Properties and structure correspond to those of graphite bisulfate.

For other salt-type graphite compounds with HF,  $H_2SeO_4$ ,  $H_3PO_4$  and  $H_3AsO_3$ , see the references.

**REFERENCES:** 

W. Rüdorff and U. Hofmann. Z. anorg. allg. Chem. 238, 1 (1938). W. Rüdorff. Z. anorg. allg. Chem. 254, 319 (1947).

## **Bromine Graphite**

Prepared by action of bromine upon graphite.

Coarse or fine crystalline graphite, in a weighing bottle, is allowed to stand over bromine in a desiccator for many hours at room temperature. The maximum weight gain of the graphite is 0.82-0.84 g. per g. of C. PROPERTIES:

The composition corresponds to an atomic ratio C:Br of about 8:1.

Black to blue-gray crystals, as in the starting graphite, but enlarged. Most of the  $Br_2$  taken up is desorbed in air and can be washed out with water.

Structure: Graphite lattice with a bromine layer in every second layer vacancy.

REFERENCE:

W. Rüdorff. Z. anorg. allg. Chem. 245, 383 (1941).

## **Metal Halide Graphite Compounds**

Prepared from graphite and anhydrous metal halides, e.g.,  $FeCl_3$ ,  $AlCl_3$ ,  $GaCl_3$ ,  $InCl_3$ ,  $MoCl_5$ ,  $UCl_5$ , etc. With the exception of  $FeCl_3$ , these metal halides can be intercalated only in the presence of free chlorine.

#### A) IRON (III) CHLORIDE GRAPHITE

A glass tube of about 2 cm. I.D. is used. At least 5 g. of anhydrous  $FeCl_3$  is sublimed onto 2 g. of coarse or fine crystalline graphite. The tube is sealed off on both sides and heated for 24 hours in an electric tubular furnace at a constant temperature of 200-300°C. One end of the tube is then withdrawn from the furnace, and the heating is continued at the same temperature until no further uptake of the desublimed  $FeCl_3$  can be observed.

PROPERTIES:

Black, highly enlarged crystals. The FeCl<sub>3</sub> content depends upon the temperature of desublimation and is in the range of 60-70%FeCl<sub>3</sub>. When heating is rapid, there is pronounced swelling with elimination of FeCl<sub>3</sub>. The crystals release FeCl<sub>3</sub> in water, dilute acids, alcohol or benzene. Depending on the size of the crystals, the washed preparations contain 52-56% FeCl<sub>3</sub>.

## B) ALUMINUM CHLORIDE GRAPHITE

Preparation is similar to that presented above under (A). About 3-4 g. of  $AlCl_3$  is sublimed in a stream of dry  $Cl_2$  onto 1 g. of graphite. The sealed reaction tube (capacity about 40 cm<sup>2</sup>) must

644

be well filled with  $Cl_2$ . The heating of the reaction mixture and subliming of the excess AlCl<sub>3</sub> are carried out at 150 to 200°C.

PROPERTIES:

Very dark blue, lustrous, highly enlarged crystals; releases  $AlCl_3$  and  $Cl_2$  when heated above  $260^{\circ}C$ ; very hygroscopic. Water and organic solvents dissolve out much of the intercalated  $AlCl_3$ . Precipitates iodine when added to KI solution. The compound corresponds approximately to  $C_{30}^+AlCl_4^- \cdot 2AlCl_3$ .

Structure of A and B. Graphite lattice expanded in the c direction with a layer of metal halide in each layer vacancy.

**REFERENCES:** 

W. Rüdorff and H. Schulz. Z. anorg. allg. Chem. <u>245</u>, 121 (1940).
 W. Rüdorff and R. Zeller. Z. anorg. allg. Chem. <u>279</u>, 182 (1955).
 W. Rüdorff and A. Landel. Z. anorg. allg. Chem. <u>293</u>, 327 (1958).

c) Volatile Carbon Compounds O. GLEMSER

#### Carbon Monoxide

СО

I. FROM FORMIC ACID

# $\begin{array}{l} \text{HCOOH} = \text{CO} + \text{H}_2\text{O} \\ \text{46.0} \quad 22.4\,l \end{array}$

A gas generator (one-liter round-bottom flask with ground glass dropping funnel and gas outlet tube) two-thirds full of concentrated phosphoric acid is heated to  $80^{\circ}$ C in a water bath. Formic acid is then added slowly, drop-by-drop. Removal of impurities (CO<sub>2</sub>, air, acid vapors, water vapor) is effected by passing the gas successively through 50% KOH solution and an alkaline solution of Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub> (25 g. of Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub> in 125 ml. of H<sub>2</sub>O, plus by 20 ml. of 70% KOH) and over KOH, CaCl<sub>2</sub> and P<sub>2</sub>O<sub>5</sub>.

Traces of  $O_2$  may also be removed with a glowing carbon filament [K. Clusius and W. Teske. Z. phys. Chem. (B) <u>6</u>, 135 (1929)]. Larger quantities of oxygen are removed in a purification train consisting of three wash bottles in series, two of which contain 100 g. of slightly amalgamated Zn and 100 ml. of  $0.1M \text{ VOSO}_4$  solution, while the third contains 100 ml. of  $H_2O$  [L. Meites and T. Meites. Anal. Chem. 20, 984 (1948)].

Very pure CO is obtained by liquefaction of the gas and double fractionation (impurities  $<10^{-3}$  mole%).

## II. FROM OXALIC ACID

$$\begin{array}{c} H_{2}C_{2}O_{4} \cdot 2 H_{2}O = CO + CO_{2} + 3 H_{2}O \\ 126.1 & 22.4 l & 22.3 l \end{array}$$

A mixture of 100 g. of oxalic acid dihydrate and 275 ml. of concentrated  $H_2SO_4$  is heated carefully in a round-bottom flask until the onset of gas generation, which must not be allowed to proceed too vigorously. The  $CO_2$ , formed in equal amounts with the CO, is absorbed in two wash bottles each containing 100 ml. of 50% KOH solution. Purification is the same as in method I.

Other preparative methods: III. From formic acid or barium formate and phosphoric acid [J. G. Thompson. Ind. Eng. Chem. 21, 389 (1929)]. This method is particularly suitable for a continuous process and gives a 92% yield.

IV. From concentrated  $H_2SO_4$  and cyanides [J. Wade and L. C. Panting. J. Chem. Soc. (London) 73, 255 (1898)].

V. By dry heating of a mixture of  $CaC_2O_4$  and CaO (A. Klemenc. Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], Vienna, 1948, p. 159).

Procedures IV and V are not as satisfactory as I.

VI. Small amounts of very pure gas may be prepared by thermal decomposition of Ni(CO)  $_4$  [A. Mittasch. Z. phys. Chem. <u>40</u>, 1 (1902); C. E. H. Bawn. Trans. Faraday Soc. 31, 440 (1935)].

VII. From CaCO<sub>3</sub> powder and Zn dust at 700 to 750°C; very pure gas results [S. Weinhouse. J. Amer. Chem. Soc. <u>70</u>, 442 (1948)].

Purification of tank CO: Possible impurities:  $O_2$ ,  $O_2$ ,  $H_2$ ,  $CH_4$ ,  $N_2$ ,  $Fe(CO)_5$ . The gas is passed slowly through a tube filled with reduced copper wire and heated to 600°C; this removes  $O_2$  and  $Fe(CO)_5$ . The activated Cu-filled tube of Meyer and Ronge (see section on Nitrogen, p. 458) may also be used. The CO then flows through a tower filled with moist KOH to absorb the  $CO_2$ . Should further purification (particularly removal of  $H_2$ ) be necessary, the CO must be liquefied and fractionally distilled.

#### PROPERTIES:

Formula weight 28.01. Poisonous, colorless and odorless gas. M.p.  $-205.1^{\circ}$ C, b.p.  $-191.5^{\circ}$ C; crit. t.  $-140.2^{\circ}$ C; crit. p. 34.5 atm.; triple pt. p. 115.0 mm. Burns with a blue flame; ignition point 700°C in the air; lower explosion limit in moist air: 12.5% CO. At elevated temperatures, decomposes into CO<sub>2</sub> and C. Solubility in  $H_2O$  (0°C) 3.3, (20°C) 2.3 vols. of CO/100 g.  $H_2O$ . About seven times as soluble in methyl and ethyl alcohols as in  $H_2O$ . Readily soluble in acetic acid, CHCl<sub>3</sub> and ethyl acetate. d (liq., -195°C) 0.814; d (gas, 0°C) 1.250 g./liter. Heat of formation -26.39 kcal./ mole.

**REFERENCES:** 

- I. E. Rupp. Chem, -Z. <u>32</u>, 983 (1908); J. Meyer. Z. Elektrochem. <u>15</u>, 506 (1929).
- II. F. Schacherl. Pub. Fac. Sc. Univ. Masaryk <u>99</u>, 5 (1928); A. Klemenc. Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], Vienna, 1948, p. 160.

## Carbon Dioxide

#### CO<sub>2</sub>

#### PURIFICATION OF TANK CO2

Possible impurities: water vapor, CO,  $O_2$  and  $N_2$ , sometimes accompanied by traces of  $H_2S$  and  $SO_2$ .

I. Moderately Pure CO<sub>2</sub>: The gas is passed sequentially through the following vessels: 1) two wash bottles with Cr(II) acetate solution or VOSO<sub>4</sub> solution to remove the bulk of the O<sub>2</sub> (also see CO, method I); 2) a U tube containing small pieces of KHCO<sub>3</sub> (for removal of acid vapors); 3) a U tube with pumice impregnated with CuSO<sub>4</sub> solution, or a wash bottle filled with 1M KMnO<sub>4</sub> solution or 1M K<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub> solution (removal of H<sub>2</sub>S); 4) a wash bottle with concentrated sulfuric acid (for drying); and 5) a Meyer and Ronge tube filled with activated Cu and Cu<sub>2</sub>O heated to 200°C (see description of the purification of N<sub>2</sub>, p. 458) for removal of the last traces of O<sub>2</sub> and CO. Instead of 3 and 4, R. Weber recommends [Angew. Chem. <u>65</u>, 136 (1953)] that organic compounds and H<sub>2</sub>S be removed by passing the gas through a mixture of 100 volumes of H<sub>2</sub>SO<sub>4</sub> (d 1.84) and 3.3 volumes of aqueous formaldehyde (40 vol. %).

If a given impurity is not present, the corresponding purification step may be omitted.

II. Purer  $CO_2$ : The gas is passed through saturated  $CuSO_4$  through KHCO<sub>3</sub> solution, and finally through a fractionator (Klemenc and Bankowski). This apparatus is a portion of the equipment presented in Fig. 153 (p. 345) for the preparation of very pure  $H_2S$ . To fractionate  $CO_2$  one employs only the section to the right of wash bottle 4, consisting of eight well-cooled U tubes and two low-temperature traps. Immediately before the last condensation trap  $A_2$ , a mercury manometer tube is inserted on a side tube. The  $CO_2$  is frozen out in  $A_1$  after passing through the first four

U tubes, cooled to the temperatures indicated in the figure. When  $A_1$  is sufficiently full, stopcock a is opened, the tube is sealed off at point c, and the remaining apparatus is evacuated to a low pressure. After the remaining four U tubes have been cooled to  $-78^{\circ}$ C (Dry Ice-acetone), the liquid nitrogen is removed from around  $A_1$ , the first fraction of gas is siphoned off, and condensation vessel  $A_2$  is immersed in liquid nitrogen. The middle fraction is collected in  $A_2$ , the residue being left in  $A_1$ . The material is sub-limed twice from  $A_2$  and the purity of the gas is checked by its vapor pressure at various temperatures. The gas is stored in 25-liter glass flasks which have been degassed by heating in high vacuum at  $350^{\circ}$ C for many hours.

SYNONYM:

Carbonic acid anhydride.

PROPERTIES:

Formula weight 44.01. Colorless, odorless gas. Subl.t. -78.48°C (atmospheric pressure); m.p. -56.7°C (5 atm.); crit. t. 31.3°C; crit. p. 72.9 atm.; crit. d 0.464; triple pt. -56.6°C at 5.11 atm. Vapor pressure (-120°C) 10.5; (-100°C) 104.2; (-82°C) 569.1 mm. Solubility in H<sub>2</sub>O. (0°C) 171; (20°C) 88; (60°C) 36 ml. of

 $CO_2/100 \text{ g}$  of  $H_2O_2$ 

d (vapor, 0°C) 1.977 g./liter; d (liq., 0°C) 0.914 (34.4 atm.); d (solid, -56.6°C) 1.512. Heat of formation -94.05 kcal./mole.

**REFERENCES:** 

- I. L. Moser. Z. anorg. allg. Chem. <u>110</u>, 125 (1921).
- II. A. Klemenc and O. Bankowski. Z. anorg. allg. Chem. 208, 348 (1932); 209, 225 (1932).

## Tricarbon Dioxide

#### $C_3O_2$

I. THERMAL DECOMPOSITION OF MALONIC ACID IN THE PRESENCE OF  $P_2O_5$ 

$$\begin{array}{c} C_{3}H_{4}O_{4} = C_{3}O_{2} + 2 H_{2}O \\ 114.1 & 68.0 \end{array}$$

Flask  $\alpha$  of the apparatus illustrated in Fig. 209 is charged with 20 g. of malonic acid, 40 g. of calcined sand and 200 g. of fresh, uncaked and well mixed  $P_2O_5$ . The system is evacuated to 0.1 mm.,

648

stopcock h is closed, and the apparatus is left to stand for some hours to complete the drying and to test for leaks. The pump is then started, stopcock m is opened once again, d is cooled with liquid nitrogen and a is heated on an oil bath to  $140^{\circ}$ C. At this temperature, decomposition is complete within about an hour and impure  $C_3O_2$  condenses in d. Now the oil bath is removed, m is closed, the pump is stopped, dry air is introduced at k, and a is removed from the system and sealed off at c. The system is again evacuated, m is closed, and the contents of d are slowly distilled into trap h, cooled with liquid nitrogen. Plugging of h should not be allowed to occur. Acetic acid and other impurities are absorbed



Fig. 209. Preparation of tricarbon dioxide from malonic acid. a) one-liter reaction flask; b) rubber stoppers, lightly greased;
c) connecting tube, 10 mm. I.D. d, h and l) traps; e) drying tower containing pea-sized, freshly calcined pieces of CaCO; g) 800-mm.-long manometer tube.

in the lime tower and the material is fractionated in high vacuum with stopcock f closed. This is done by placing h in an alcohol bath (-110°C to -115°C), while l is cooled with liquid nitrogen, stopcock i is closed, and the condensate in h is melted. Then h is again immersed in the alcohol bath, i is opened, and distillation into l proceeds. Pressure changes are followed on the manometer. When the manometer pressure is still but a few tenths of a millimeter, the vapor pressure is determined at 0°C and compared with that of the pure gas (573.5 mm.).

Separation of the  $C_3O_2-CO_2$  mixture ( $CO_2$  is a product of the side reaction  $C_3H_4O_4 = CH_3COOH + CO_2$ ) is difficult. Therefore, when the solid phase ( $CO_2$ ) has disappeared, it is advisable to reduce the bath temperature to  $-125^{\circ}C$  or  $-130^{\circ}C$  for completion of the separation. Fractionation takes about 15 hours (A. Klemenc, loc. cit.).

A. Klemenc, R. Wechsberg, and G. Wagner have suggested inserting before the lime tower a tube filled with glass wool to retain the  $P_2O_5$ , condensing the reaction products ahead of stop-cock f in a liquid-nitrogen-cooled vessel, and finally distilling once again into h via a tube filled with glass wool and a lime tower.

The yield of  $C_3O_2$  is 2.94 g. (22%). Determination of yield:  $C_3O_2$  (alone or mixed with other gases) is passed through a solution of aniline in xylene. The malonanilide formed is virtually insoluble in xylene and precipitates as colorless crystals, m.p. 223°C.

Additional preparative method: Thermal decomposition of diacetyltartaric anhydride (Klemenc et al.). The yield is better, but the  $C_3O_2$  contains ketene, which cannot be completely separated.

SYNONYM:

Carbon suboxide.

PROPERTIES:

Formula weight 68.03. Colorless, highly refractive liquid or colorless, poisonous gas of stifling odor. M.p.  $-112.5^{\circ}$ C, b.p. 6.7°C; d<sup>4</sup><sub>4</sub> 1.114; vapor pressure (0°C) 573.5 mm. (corr.).

Attacks hydrocarbon but not silicone grease. The gas can be stored at pressures of up to 100 mm., but it is common even at these pressures for polymerization to occur, giving a red, water-soluble product. This invariably occurs at higher pressures or in the liquid state. The presence of  $P_2O_5$  facilitates polymerization. Decomposes when passed through heated glass tubes, forming a mirror surface.

Soluble in  $CS_2$  and xylene. Quantitatively decomposed by water (within one hour) to malonic acid. Forms malonamide with ammonia. Heat of formation 47.4 kcal./mole.

REFERENCES:

A. Stock and H. Stoltzenberg. Ber. dtsch. chem. Ges. 50, 498 (1917).

A. Klemenc. Die Behandlung und Reindarstellung von Gasen [Treatment and Purification of Gases], 2nd ed., Vienna, 1948, p. 164.

A. Klemenc, R. Wechsberg and G. Wagner. Monatsh. Chem. <u>66</u>, 337 (1935).

## Carbonyl Chloride

## $\operatorname{COCl}_2$

#### PURIFICATION OF TANK COCl<sub>2</sub>

Possible impurities:  $CO_2$ , CO, air, HCl and H<sub>2</sub>O, total approximately 1%.

I. Small quantities: The  $COCl_2$  is removed from the tank and condensed with an ice-salt mixture. To remove the volatile portion, about a fifth of the condensate is permitted to evaporate, and the residue is fractionated in high vacuum until all fractions have the same vapor pressure. The pure gas has a vapor pressure  $(0^{\circ}C)$  of 556.5 mm.

II. Larger quantities, moderate purity requirements: Carbonyl chloride is condensed in a flask with sealed-in gas inlet tube and cooled with an ice-salt mixture. A water-cooled bulb-type reflux condenser is placed on top of the flask. The condenser is connected to a Hempel gas burette filled with  $7-8^{\circ}C$  water. The burette is joined to a downward condenser (cooled with ice-salt), which in turn is connected to a receiver. The cooling mixture is removed from around the flask, about a fifth of the liquid is permitted to evaporate, and the receiver is cooled, thus condensing the COCl<sub>2</sub>. The impurity content is less than 0.4%.

Analytical determination: Carbonyl chloride is shaken with an aqueous aniline solution and the diphenylurea product is determined gravimetrically or, after conversion to  $NH_3$ , colorimetrically.

SYNONYM:

Phosgene.

PROPERTIES:

Formula weight 98.92. Colorless, highly poisonous gas of stifling odor reminiscent of rotten hay.

M.p. -128°C, b.p. 7.5°C; crit. t. 181.7°C; crit. p. 55.3 atm; vapor pressure (0°C) 556.5 mm.

Strongly attacks stopcock grease; may be stored in glass vessels at 0°C. Very slightly soluble in cold water. Hot water hydrolyzes it readily to HCl and CO<sub>2</sub>. Readily soluble in benzene, toluene, glacial acetic acid, CCl<sub>4</sub> and hexamethylenetetramine, as well as in AsCl<sub>3</sub> and S<sub>2</sub>Cl<sub>2</sub>. Decomposes to CO and Cl<sub>2</sub> on heating (503°C, 50%; 800°C, 100%).

d (liq.,  $0^{\circ}$ C) 1.436; vapor d (18.6°C) 1.392 g./liter. Heat of formation -53.3 kcal./mole.

**REFERENCES:** 

- I. A. Stock and E. Wustrow. Z. anorg. allg. Chem. <u>147</u>, 245 (1925).
- II. E. Paternò and A. Mazzuchelli. Gazz. Chim. Ital. <u>50</u>, 30 (1920).

## Carbon Disulfide

#### $CS_2$

#### PURIFICATION OF COMMERCIAL MATERIAL

Possible impurities: dissolved S,  $H_2S$ ,  $H_2SO_3$ ,  $H_2SO_4$ , organic sulfur compounds,  $H_2O_4$ .

Half a liter of  $CS_2$  is shaken for an hour with 100-200 g. of Hg and some  $P_2O_5$ ; the mixture is filtered and the filtrate is distilled in a column, in the dark if possible. The low-boiling fraction and the high-boiling fraction (recognizable by the yellow color of the residue in the distillation flask) are discarded. The bulk of the distillate is shaken once again with Hg and  $P_2O_5$  and fractionally distilled, and the first and last cuts are discarded. These steps are repeated until no black HgS is formed. If a very pure product is desired, distillation in high vacuum is necessary. If the only impurity is  $H_2S$ , a single fractionation is sufficient.

Testing for purity: Residue: When concentrated by evaporation on the water bath, 50 ml. of  $CS_2$  should leave no residue. Dissolved S: When thoroughly agitated with dry Hg in a dry vessel, the Hg must not acquire a dark coating. H<sub>2</sub>S: No brown tint should appear on shaking with lead carbonate. For H<sub>2</sub>SO<sub>3</sub> and H<sub>2</sub>SO<sub>4</sub>: If H<sub>2</sub>O shaken with the CS<sub>2</sub> gives an acid reaction, either H<sub>2</sub>SO<sub>3</sub> or H<sub>2</sub>SO<sub>4</sub> is present.

SYNONYMS:

Carbon bisulfide, dithiocarbonic anhydride.

**PROPERTIES:** 

Formula weight 76.13. The liquid is water clear and highly refractive, with an odor of ether. F.p.  $-111.6^{\circ}$ C, m.p.  $-108.6^{\circ}$ C, b.p. 46.25°C; crit. t. 273.05°C;  $n_D^{18}$  1.6295; vapor pressure (0°C) 127.3 mm.

Decomposes on standing for a long time, especially in light. Decomposition is recognized by the unpleasant odor. Burns with a blue flame to  $CO_2$  and  $SO_2$ ; ignition point 236°C.

Slightly soluble in  $H_2O$ ; soluble in ethyl alcohol, benzene, ether, and essential and aliphatic oils.

 $d_4^\circ$  1.293,  $d_4^{2\circ}$  1.262. Heat of formation 21.0 kcal./mole.

**REFERENCES:** 

- A. Stock, A. Brandt and H. Fischer. Ber. dtsch. chem. Ges. <u>58</u>, 643 (1925).
- L. Vanino. Handb. d. präp. Chem., 2nd ed., Vol. 1, Stuttgart, 1921.

## Tricarbon Disulfide

#### $C_3S_2$

Prepared by reaction of Zn with liquid  $CS_2$  in an electric arc:

 $3 CS_2 + 4 Zn = C_3S_2 + 4 ZnS$ 228.4 261.5 100.2 389.8

Chemically pure CS<sub>2</sub> (1500 ml.), previously agitated with Hg and  $P_2O_5$  and then fractionated, is placed in a two-liter round-bottom flask immersed in ice up to the neck. A three-hole stopper-for CO<sub>2</sub> inlet tube and for the two 6-mm. brass wires of the electrode holder-is fitted loosely into the flask opening. The cylindrical zinc anode (15-20 mm, long, 6 mm, diameter) is supported on a brass wire with two right-angle bends and is placed just above the bottom of the flask. Opposite it is the graphite cathode (80 mm. long, 12 mm. diameter), mounted on the other brass wire. The upper end of this brass wire is connected to the mechanical adjustment device of a differential arc torch and moved downward to prevent interruption of the arc due to vaporization of the electrodes. A stream of  $CO_2$  is now introduced and the two electrodes are brought together. The current is turned on and the electrodes are separated to strike the arc. The current should be 4-5 amp. with 20-25 volts across the terminals. The CS<sub>2</sub> soon becomes opaque due to finely divided carbon.

The reaction product, a red-brown liquid of stifling odor, is filtered, the filter residue washed with some  $CS_2$  and the filtrate shaken with 200 g. of Hg and some  $P_2O_5$ . After filtering, four fifths of the liquid is vaporized on a water bath and then evacuated at room temperature with a pump until all the  $CS_2$  is expelled. The remaining red fluid is distilled in high vacuum into a receiver cooled to  $-40^{\circ}C$ , in which the  $C_3S_2$  condenses as a yellowish red solid substance of high purity. The yield is 800-850 mg. of  $C_3S_2$ (50% of theoretical) when the arc is on for five hours.

Determination of  $C_3S_2$ : The CS<sub>2</sub> solution, concentrated to about 100 ml., is completely vacuum distilled into a cold flask, and the condensate, consisting solely of CS<sub>2</sub> and C<sub>3</sub>S<sub>2</sub>, is treated with excess Br<sub>2</sub>, allowed to stand for a few hours, and evaporated in a stream of dry air. The residue is yellow C<sub>2</sub>S<sub>2</sub>Br<sub>2</sub>, stable in air.

SYNONYM:

Carbon subsulfide.

**PROPERTIES:** 

Bright red, highly refractive liquid with a strong odor that affects the mucous membranes. M.p.  $-0.5^{\circ}C$ ; vapor pressure (50°C) 8 mm., (90°C) 48 mm.

Decomposes above 90°C. Solutions of  $C_3S_2$  in  $CS_2$  containing more than 1%  $C_3S_2$  are not stable and gradually precipitate black polymerization products. Sunlight has the same effect, even in dilute solutions. At 160°C, pure  $C_3S_2$  yields a solid black product. With aniline,  $C_3S_2$  forms thiomalonanilide.

**REFERENCES:** 

A. Stock and P. Praetorius. Ber. dtsch. chem. Ges. <u>45</u>, 3568 (1912);
A. Stock, A. Brandt and H. Fischer. Ber. dtsch. chem. Ges. <u>58</u>, 643 (1925).

## Carbonyl Sulfide

COS

CO + S = COS22.4 l 32.1 22.1 l

I.

Carbon monoxide is admitted into the apparatus shown in Fig. 210 to expel the air and pure S in c is heated to the boiling point. The side tube d is heated to  $350^{\circ}$ C by an electric furnace. The resultant COS flows, together with the CO, through vessels e, f and g, and is condensed in the trap h. Tube  $g_1$  is kept at  $-20^{\circ}$ C;  $g_2$  and  $g_3$  are at  $-60^{\circ}$ C. The condensed gas still contains some 2% of impurities, which are removed by high-vacuum distillation.

The gas rate must be so regulated that excessive quantities of S dust do not reach the vicinity of the cooled rubber stopper, or the tube will readily be plugged. The optimum flow rate is approximately eight liters/hour, in which case the yield is 75%.

II.  $2 \text{ NH}_4 \text{NCS} + 2 \text{ H}_2 \text{SO}_4 + 2 \text{ H}_2 \text{O} = 2 (\text{NH}_4)_2 \text{SO}_4 + 2 \text{ COS}_{152.2}$ 152.2 196.2 264.3 44.2 l

A cooled mixture of 2080 g. of concentrated  $H_2SO_4$  and 1000 g. of  $H_2O$  is placed in a round-bottom flask provided with a gas outlet tube and immersed in a water bath. Saturated  $NH_4NCS$ solution (200 ml.) is added from a dropping funnel. The water bath is now heated to about 30°C, resulting in a vigorous generation of gas. The solution is shaken back and forth. The evolved gas (impurities consist of  $NH_3$ ,  $H_2S$ ,  $CS_2$ ,  $CO_2$ , HCN,  $H_2O$ , etc.) passes through a ten-bulb tube filled with 33% NaOH. It is then dried in two lime-filled drying towers and in another filled with CaCl<sub>2</sub> and is finally condensed in a U tube kept at  $-70^{\circ}C$ . The yield is 75%, based on HNCS.

The gas is separated from the admixed  $CO_2(0.1 \text{ vol.}\%)$  by highvacuum fractionation. Qualitative test for  $CO_2$  in COS: reaction with Ba(OH)<sub>2</sub> solution. Pure COS shows no discernible initial reaction.

#### PROPERTIES:

Formula weight 60.07. Colorless, poisonous gas with a mild odor. M.p.  $-138.2^{\circ}$ C, b.p.  $-50.2^{\circ}$ C; vapor pressure ( $-75^{\circ}$ C) 210 mm. Mixtures of air and COS are explosive when they contain between 11.9 and 26.5 vol.% COS. At 300°C, COS decomposes to CO and S. In air, burns with a blue flame to give CO<sub>2</sub> and SO<sub>2</sub>. May be stored when dry; hydrolyzed by water and water vapor. Water reacts slowly, yielding CO<sub>2</sub> and H<sub>2</sub>S. Absorbed and rapidly decomposed by KOH. Solubility (20°C) 0.54 ml. of COS/g. of H<sub>2</sub>O; (22°C) 8 ml. of COS/ml. of alcohol; (22°C) 15 ml. of COS/ml. of toluene. Solubility in CS<sub>2</sub> is even greater. d (liq.,  $-87^{\circ}$ C) 1.24; vapor d (0°C) 1.073 g./ liter. Heat of formation -33.9 kcal./mole.



Fig. 210. Preparation of carbonyl sulfide.  $\alpha$ ) manometertype flowmeter; b) safety valve; c) reaction vessel; d) reaction tube, 400 mm. long and 25 mm. I.D., filled with pea-sized pieces of pumice, the top end wrapped with a water-cooled lead coil and closed with a rubber stopper; e) U tube with copper turnings; f) wash bottle with 33% NaOH, cooled with ice water; g) traps; h) trap cooled with liquid nitrogen, preceded by a tube constriction for sealing off.

REFERENCES:

- I. A. Stock and E. Kuss. Ber. dtsch. chem. Ges. <u>50</u>, 159 (1917);
   A. Klemenc. Z. anorg. allg. Chem. 191, 246 (1930).
- II. P. Klason. J. prakt. Chem. (2) 36, 67 (1887).

## **Carbonyl Selenide**

COSe

# CO + Se = COSe $22.4 l \quad 79.0 \quad 22.2 l$

Carbon monoxide, generated from formic and phosphoric acids, is passed through 50% KOH solution, dried with KOH, CaCl<sub>2</sub> and

 $P_2O_5$ , and passed through a flow meter into a Vycor or ceramic reaction tube in which Se is heated to  $780^{\circ}C$ . Because of the Se dust, the apparatus shown in Fig. 210 is used. However, section d is not filled with pumice and is not heated. The gas then passes through the various traps, which are cooled with ice, Dry Ice and liquid nitrogen. The condensate from the trap cooled with liquid nitrogen is fractionated in high vacuum. The yield at 15 liters of CO/hour is about 9.8 vol.% COSe.

Determination of yield: The gas mixture is introduced into a glass bulb of known volume and hydrolyzed for 15 minutes with 2N NaOH; air is then admitted while the solution is heated and the precipitated Se is weighed after washing with water and methyl alcohol and drying at 105°C.

Analysis: A weighed amount of COSe is absorbed in KOBr solution. At the end of the reaction, concentrated hydrochloric acid is added until the solution is mildly acidic. It is then reduced to elemental Se at 70°C by addition of hydrazine sulfate. After cooling, the material is filtered, washed with water and methanol, dried at  $105^{\circ}$ C and reweighed.

Other preparative method:  $Al_2Se_3 + 3COCl_2 = 2AlCl_3 + 3COSe$ (O. Glemser and T. Risler).

PROPERTIES:

Formula weight 106.97. Colorless, very poisonous gas, with a characteristic odor reminiscent of H<sub>2</sub>Se. M.p.  $-124.4^{\circ}$ C, b.p.  $-21.7^{\circ}$ C; vapor pressure ( $-31.4^{\circ}$ C) 498.7 mm.; crit. t.  $121.1^{\circ}$ C.

Hydrolyzes with water or water vapor, precipitating red Se. Acids act slowly, while oxidizing acids and  $H_2O_2$  oxidize it to SeO<sub>3</sub><sup>2</sup>-. Alkaline solutions hydrolyze it rapidly and quantitatively to Se<sup>2</sup>- and CO<sub>3</sub><sup>2</sup>-. Decomposed into CO and Se by porous substances such as activated charcoal. Very soluble in COCl<sub>2</sub>.

d (liq., 41°C) 1.812.

REFERENCES:

- T. G. Pearson and P. L. Robinson. J. Chem. Soc. (London) <u>1932</u>, 652.
- O. Glemser and T. Risler. Z. Naturforsch. <u>3 b</u>, 1 (1948).

## Carbon Diselenide

CSe<sub>2</sub>

I.

 $\begin{array}{c} CH_{2}Cl_{2}+2\,Se=CSe_{2}+2\,HCl\\ 85.0 & 157.9 & 169.9 \end{array}$ 

As shown in Fig. 211, a dry stream of  $N_2$  is saturated with  $CH_2Cl_2$  vapor, and the gas mixture is introduced into a Vycor

flask in which Se is heated to  $550-600^{\circ}$ C. The crude product precipitates in the receiver, equipped with a Liebig condenser. The cooling by the condenser is stopped and the liquid is forced with steam into a cooled flask, separated from the H<sub>2</sub>O and dried over CaCl<sub>2</sub>. It is then distilled at  $46^{\circ}$ C and 50 mm. in a fractionating column (30 cm. long).



fused Se

Fig. 211. Preparation of carbon diselenide.

A deposit of C forms in the reaction zone. This may inhibit the reaction under certain conditions. This deposit can be prevented by adding the Se in smaller portions.

The yield is greater if the reaction vessel employed is of the same type as that used to produce COS(Fig. 210) and if the inclined tube d, which is empty in this experiment, is heated to  $200^{\circ}C$  with an electric furnace. The entraining of red Se dust by the gas stream is thus greatly decreased. The yield is 52%, based on the Se charged.

II.

$$\begin{array}{c} \text{CCl}_4 + 2 \text{ H}_2\text{Se} = \text{CSe}_2 + 4 \text{ HCl} \\ 153.9 & 44.8 l & 169.9 \end{array}$$

A stream of oxygen-free  $N_2$  is passed into a wash bottle equipped with a glass frit and filled with  $H_2O$ . The bottle is held at  $64^{\circ}C$ (p = 180 mm.) and the gas mixture is run, without condensing the vapor, through a tube filled with finely divided  $Al_2Se_3$  and pumice. The outlet mixture of  $N_2$  and  $H_2Se$  gases is dried with  $CaCl_2$ . A stream of  $N_2$  is saturated with  $CCl_4$  in the same manner in a second wash bottle ( $20^{\circ}C$ , 90 mm.). The rate of the two gas streams is six liters/hour. The two streams are combined in a 500-mm.-long Pyrex tube heated to 500°C, and then passed through two traps held at  $-70^{\circ}$ C (Dry Ice-acetone). At the end of the reaction, H<sub>2</sub>O is added to the condensate, and CCl<sub>4</sub> and CSe<sub>2</sub> are distilled from the higher boiling impurities (e.g., selenium chlorides). The distillate is fractionated in an efficient column, first at atmospheric pressure, then at 125 mm. At 68°C virtually pure CSe<sub>2</sub> distills over. This is distilled further.

SYNONYM:

Carbon selenide.

PROPERTIES:

Golden yellow, highly refractive liquid; odor of rotten radishes. M.p.  $-45.5^{\circ}$ C, b.p. 125-126°C (760 mm.); vapor pressure (0°C) 4.7 mm.;  $n_{D}^{20}$  1.845.

Very sensitive to light; turns brown and finally black on standing. Burns with difficulty. When heated in a sealed tube to  $150^{\circ}$ C, solidifies to a black mass (polymerizes?). Insoluble in H<sub>2</sub>O; dissolves, yielding a yellow liquid, in CS<sub>2</sub>, CCl<sub>4</sub>, ether, benzene, nitrobenzene, dioxane, ethyl acetate and acetone. Slightly soluble in glacial acetic acid and alcohol, decomposing these rapidly; pyridine behaves in the same manner. Dissolves copious quantities of flowers of sulfur, but red Se hardly at all. Decomposed on boiling with concentrated nitric acid. Decomposed by concentrated NaOH to a marked degree, but only on heating. d<sup>2</sup><sub>4</sub> 2.682. Heat of formation 34 kcal./mole.

**REFERENCES:** 

- I. I. G. Ives, R. W. Pittman and W. Wardlaw. J. Chem. Soc. (London) 1947, 1080.
- II. H. G. Grimm and H. Metzger. Ber. dtsch. chem. Ges. <u>69</u>, 1356 (1936).

## Hydrogen Cyanide

## HCN

I.

$$2 \operatorname{NaCN}_{98.0} + \operatorname{H}_2 \operatorname{SO}_4 = \operatorname{Na}_2 \operatorname{SO}_4 + 2 \operatorname{HCN}_{98.0}$$
98.0 98.1 142.1 54.1

The long-neck, round-bottom flask a of the apparatus in Fig. 212, containing 1 kg. of concentrated sulfuric acid, 400 ml. of H<sub>2</sub>O, 20 g. of FeSO<sub>4</sub> and a few boiling stones, is heated on a water bath to 90°C and a solution of 1 kg. of commercial NaCN in 1.2 liters



Fig. 212. Preparation of hydrogen cyanide. a) five-liter generating flask heated to 90°C; b) dropping funnel; c) condenser; d and e) filter flasks; f) ground joint flask.

of  $H_2O$  is carefully added dropwise. The decomposition ends in 1-1.5 hours. The water bath is then brought to a boil; the HCN is completely driven off within 30 minutes. Water vapor is then removed in condenser c, in filter flask d containing 20 ml. of 2N sulfuric acid and some boiling stones, and in the second filter flask e, containing 200 g. of CaCl<sub>2</sub> over a layer of glass wool. Both the latter and the condenser are heated to 50°C. Hydrogen cyanide is condenser in the ground joint flask f, which is cooled with ice-salt mixture. A second condensation flask is attached to the first for safety. The yield is 550 g.

II.

$$\begin{array}{c} K_4[Fe(CN)_6] + 3 H_2SO_4 = 2 K_2SO_4 + FeSO_4 + 6 HCN \\ (3 H_2O) \\ 422.4 \\ 294.2 \\ 348.5 \\ 151.9 \\ 162.2 \end{array}$$

A 200-g. portion of  $K_4[Fe(CN)_6] \cdot 3H_2O$ , not too finely crushed, is placed in a round-bottom two-liter flask and a cold mixture of 160 g. of concentrated  $H_2SO_4$  and 250 g. of  $H_2O$  is added. The flask is then sealed with a well-seated rubber stopper, through which a 400 mm.-long glass tube of 10 mm. I.D. is inserted. The upper end of the tube connects to a narrow tube, which in turn is connected to three CaCl<sub>2</sub> tubes placed in series below it. These tubes are immersed up to their necks in a 40°C water bath. The last tube has a three-way stopcock which permits discharge to the hood. This stopcock is connected to a mercury check valve, which in turn is connected to a Liebig condenser, the bottom end of which is connected to a well-cooled round-bottom flask by means of a two-hole rubber stopper. An outlet to the hood passes through the other hole.
The flask should be gently heated (sand bath or asbestos plate). The HCN condenses in the Liebig condenser and in the round-bottom flask.

Purification is accomplished by three further distillations, discarding each time the first and last fractions. The material is then fractionated in a vacuum column. High vacuum is used when maximum purity is required.

Storage of HCN: Two drops of concentrated hydrochloric acid are added to the flask, which is then stored in an ice chest. The stopper of the flask should be secured with a wire lock.

Other preparative method: From Hg(CN)<sub>2</sub> and H<sub>2</sub>S [J. R. Partington and M. F. Caroll. Phil. Mag. (6) <u>49</u>, 665 (1925)].

SYNONYMS:

Hydrocyanic acid, formyl nitrile, prussic acid.

PROPERTIES:

Formula weight 27.03. Colorless, very poisonous gas; odor of bitter almonds. M.p.  $-13.24^{\circ}$ C, b.p. 25.70°C; crit. t. 183.5°C; crit. p. 55 atm.; triple pt. p. 140 mm.; n<sup>10</sup><sub>D</sub> 1.2675.

Burns with a red-blue flame. Miscible with water, alcohol and ether in all proportions. Weak acid.

d (0°C) 0.715, d (liq., 18°C) 0.691. Heat of formation 30.7 kcal./ mole.

**REFERENCES:** 

I. K. H. Slotta. Ber. dtsch. chem. Ges. 67, 1028 (1934).

II. L. Gattermann. Liebigs Ann. Chem. 357, 318 (1907).

#### Cyanogen

#### $(CN)_2$

I.

$2 \operatorname{CuSO}_4$ -	+ 4 KCN =	= 2 CuCN	+ 2 K <sub>2</sub> SO <sub>4</sub> ·	$+ (CN)_{2}$
$(5 H_2 O)$				
499.4	260.5	179.2	348.5	22.4 l

Potassium cyanide solution is permitted to drip on 500 g. of finely pulverized  $CuSO_4 \cdot 5H_2O$  placed in a two-liter round-bottom flask with a two-hole stopper for a dropping funnel and gas outlet tube. The rate of addition is determined by the desired rate of gas evolution. If the rate of evolution of  $(CN)_2$  becomes too low, the flask is heated on a water bath. The  $(CN)_2$  passes through an empty, ice-cooled wash bottle and then a CaCl<sub>2</sub> tube and is condensed in a receiver held at  $-55^{\circ}C$ . To regenerate the CuCN formed in the flask, the liquid is decanted (after the evolution of gas has ceased) and about 1.2 liters of FeCl<sub>3</sub> solution (d 1.26) is added to the moist cyanide, after which further (CN)<sub>2</sub> may be generated.

For purification,  $(CN)_2$  is passed through an evacuated  $P_2O_5$  tube (300 mm. long, 30 mm. I.D.) and condensed in a flask cooled with liquid nitrogen. The condensate may be redistilled in high vacuum.

II. PREPARATION BY THERMAL DECOMPOSITION OF AgCN

A) SILVER CYANIDE AgCN

 $\begin{array}{c} AgNO_{3} + KCN = AgCN + KNO_{3} \\ 169.9 & 65.1 & 133.9 & 101.1 \end{array}$ 

A cold, saturated solution of  $AgNO_3$  is precipitated with the stoichiometric quantity of 78% solution of KCN; the AgCN is rapidly filtered and immediately heated with ammonia (d 0.88). The AgCN that precipitates on cooling is twice recrystallized from ammonia in the same manner and is then dried for four days at 140°C to remove NH<sub>3</sub> and H<sub>2</sub>O.

Pale brown powder, stable to light, very slightly soluble in acids.

B) CYANOGEN (CN)<sub>2</sub>

 $2 \operatorname{AgCN}_{267.8} = (\operatorname{CN})_2 + 2 \operatorname{Ag}_{215.8}$ 

Pulverized AgCN is charged into a Vycor tube connected to a high-vacuum apparatus. The substance is first outgassed in high vacuum at  $280-330^{\circ}$ C and then heated to  $330-380^{\circ}$ C, which results in its decomposition and generation of  $(CN)_2$ . The gas is passed through a  $P_2O_5$  drying tube and condensed in a receiver cooled with liquid nitrogen. The product, already relatively pure, is refractionated in high vacuum.

Other preparative method: Thermal decomposition of  $Hg(CN)_2$ . However, the use of AgCN is more advantageous, because the  $Hg(CN)_2$  readily sublimes without releasing  $(CN)_2$  [J. H. Perry and D. C. Bardwell. J. Amer. Chem. Soc. <u>47</u>, 2629 (1925)].

SYNONYMS:

Dicyanogen, oxalic acid dinitrile.

PROPERTIES:

Formula weight 52.04. Colorless, poisonous, lachrymatory gas with a stifling odor. M.p.  $-27.83^{\circ}$ C, b.p.  $-21.15^{\circ}$ C; crit. t. 218.30°C; crit. p. 59.75 atm.

Burns with a peach-blossom-colored, blue-edged flame; mixtures containing 14 vol.% O<sub>2</sub> are explosive.

Soluble in  $H_2O$ , alcohol and ether. The solutions quickly decompose. Polymerizes to solid, brown-black paracyanogen on heating or even in sunlight. Forms HCN and HNCO ("pseudohalogen") with water.

d (b.p.) 0.954. Heat of formation 62.0 kcal./mole.

#### **REFERENCES:**

- I. J. McMorris and R. M. Badger. J. Amer. Chem. Soc. <u>55</u>, 1954 (1933).
- II. R. P. Cook and P. L. Robinson. J. Chem. Soc. (London) <u>1935</u>, 1001.

## Cyanogen Chloride

## **CNC**l

I.

 $\frac{K_2[Zn(CN)_4] + 4 Cl_2 = 4 CNCl + 2 KCl + ZnCl_2}{247.6} \frac{283.6}{245.8} \frac{245.8}{149.1} \frac{149.1}{136.3}$ 

A solution of 130 g. of KCN in 200 ml. of water is added to a solution of 145 g. of  $ZnSO_4 \cdot 7 H_2O$  in 200 ml. of water held in flask *a* (see Fig. 213). This results in a suspension of  $K_2[Zn(CN)_4]$  in 400 ml. of water. The suspension is vigorously stirred with a ground glass stirrer, the apparatus is purged with N<sub>2</sub>, and  $Cl_2$  is introduced through a fritted glass filter. At a  $Cl_2$  rate of 8 to 10 bubbles per second, a steady, fast stream of CNCl is produced 1-1.5 hours after the start of the run. Prior to gas generation the mixture evolves some heat. This is removed by cooling with running water so as to keep the temperature below 20°C. Foam is broken up by the stirrer in the broad upper section of the reaction flask. The product is dried over  $CaCl_2$  and is frozen out in a receiver cooled with an ice-salt mixture or, even better, with Dry Ice-acetone. After one half of the required quantity has



Fig. 213. Preparation of cyanogen chloride.

been introduced the  $Cl_2$  stream rate is reduced every hour by 2 to 3 bubbles per second, so that no unreacted  $Cl_2$  can contaminate the product. After five hours the reaction mixture clears up. The  $Cl_2$  flow is interrupted and the residual CNCl is driven off with  $N_2$ .

The yield of pure CNCl is 98%, based on KCN, and 85% based on  $Cl_2$ . The product requires no further purification. It is entirely free of chlorine. The content of possible impurities other than chlorine is less than 0.1%. The cation bound to the  $[Zn(CN_4)]^{2-}$  complex is not important. Experiments with Na<sub>2</sub>[Zn(CN)<sub>4</sub>] and Ca[Zn(CN)<sub>4</sub>] result in equally satisfactory yields and equally pure products.

NaCN	$+ Cl_2 =$	NaCl	+ CNCl
49.0	70.9	58.4	61.5

II.

Pulverized NaCN (49 g.) and 170 ml. of  $CCl_4$  are charged into a 500-ml. three-neck flask (see Fig. 214), provided with a mercury seal stirrer and gas inlet and outlet tubes. The flask is cooled to -5 to  $-10^{\circ}$ C in an ice-salt mixture, and the air is displaced with N<sub>2</sub>. Now 2 ml. of glacial acetic acid is added to the reaction mixture, the stirrer is started, and  $Cl_2$  is introduced. The  $Cl_2$  flow rate is adjusted to assure its complete absorption. No gas bubbles



Fig. 214. Preparation of cyanogen chloride.

should form in the wash bottle attached in series with the apparatus. The temperature must be rigorously held at  $-5^{\circ}$ C or less, since otherwise CNCl reacts with NaCN to form  $(CN)_{X^{\circ}}$ . The reaction ends after about 4.5 hours. The chlorine flow is stopped, the receiver is cooled to  $-40^{\circ}$ C with Dry Ice-acetone, the spiral condenser is encased in an ice-salt mixture, and a slow N<sub>2</sub> stream is passed through the apparatus. The temperature of the three-neck flask is allowed to rise to  $60-65^{\circ}$ C over a period of 1-1.5 hours, so that all the CNCl distills. The Cl<sub>2</sub> dissolved in the CNCl can be removed either by placing a distillation column cooled with a  $-25^{\circ}$ C bath over the Erlenmeyer flask containing the distillate, the CNCl being permitted to reflux while gaseous Cl<sub>2</sub> escapes; or by freezing the product at  $-79^{\circ}$ C, removing the Cl<sub>2</sub> in a vacuum apparatus, and fractionating the residue. The yield is 44-47 g. (72-77%).

#### **PROPERTIES:**

Colorless liquid or colorless, lachrymatory gas. M.p.  $-6.5^{\circ}$ C, b.p. 13°C; d (4°C) 1.218. Vapor pressure (0°C) 445 mm. Attacks Hg slightly. Pure CNCl does not polymerize. Exceptionally poisonous. Therefore, all work must be done under a good hood. The experimenter is strongly advised to protect himself with a gas mask when working with CNCl. Solubility: 2.5 liters in 100 ml. of water (20°C); 10 liters in 100 ml. of alcohol (20°C); 5 liters in 100 ml. of ether (20°C).

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## Cyanogen Bromide

#### **CNBr**

I.

 $Br_2 + NaCN = NaBr + CNBr$ 159.8 49.0 102.9 105.9

One kilogram of  $Br_2$  (320 ml.) is covered with 150 ml. of water in a two-liter ground joint flask placed under a good hood. The stirrer is then turned on and a solution of 420 g. of NaCN (i.e., one third excess) in 850 ml. of water is added at the rate of 1 drop per second. The temperature of the mixture must be kept below 20°C. Any local excess of cyanide is carefully avoided because it leads to formation of (CN)<sub>x</sub>. The last 150 ml. of the NaCN solution is diluted with twice that amount of water. Further dropwise addition of the solution is best performed manually, and the flask should be vigorously shaken by hand after each addition. When a persistent brown tint appears the rest of the NaCN solution is discarded. The addition of NaCN takes about five hours.



Fig. 215. Preparation of cyanogen bromide.

As shown in Fig. 215, a large diameter tube bent into a V is attached to the round-bottom flask. The shorter arm of the tube is filled with granular  $CaCl_2$ . The flask is placed on a water bath and the CNBr is distilled. It is collected in a 750-ml. powder bottle serving as receiver. The yield of snow-white crystals is 590 g. or 90% of theoretical, based on Br. The material can be stored in this form for a long time. Brownish CNBr is not stable.

The bottles used for storage are preferably closed with corks well coated with paraffin rather than with glass stoppers.

II. A modification of the process in terms of the quantities required is reported by Hartmann and Dreger. Half a kilogram of  $Br_2$  is covered with 50 ml. of water and treated with a solution of 170 g. of NaCN (i.e., about 1/8 more than the stoichiometric) in 1200 ml. of water (i.e., 2.5 times as much as in method I). The yield is 73-85%.

III. Other preparative method: To avoid loss of half the Br<sub>2</sub> in byproduct NaBr, Zmaczyński recommends the reaction:

 $2 \text{ KCN} + \frac{1}{2} \text{ H}_2 \text{SO}_4 + \text{Br}_2 + \text{Cl}_2 = \frac{1}{2} \text{ K}_2 \text{SO}_4 + \text{ KCl} + \text{HCl} + 2 \text{ CNBr}$ 

The stoichiometric quantities of  $H_2SO_4$  and  $Br_2$  are added to a KCN or NaCN solution. At the end of the reaction the same quantity of KCN or NaCN is added and  $Cl_2$  is introduced until one drop of the solution causes starch-iodide paper to turn blue. The temperature must be held at 0 to  $+5^{\circ}C$  during the entire process. Further treatment follows that described in method I.

#### **PROPERTIES:**

Formula weight 105.93. Colorless needles. M.p.  $52^{\circ}$ C, b.p.  $61.6^{\circ}$ C; d ( $20^{\circ}$ C) 2.01. Soluble in ether. Because of the high toxicity of the product, all work must be done under an efficient hood. The experimenter should wear a gas mask.

#### REFERENCES:

- I. K. H. Slotta. Ber. dtsch. chem. Ges. 67, 1029 (1934).
- II. W. W. Hartmann and E. E. Dreger in: Organic Syntheses, Coll. Vol. 2, p. 150, New York, 1948.
- III. E. Zmaczyński. Ber. dtsch. chem. Ges 59, 711 (1926).

#### Cyanogen lodide

#### CNI

I.

 $\begin{array}{c} NaCN + I_2 = CNI + NaI \\ 49.0 \quad 253.8 \quad 152.9 \quad 149.9 \end{array}$ 

A solution of 27 g. (0.55 mole) of NaCN in 100 ml. of water is allowed to cool to 0°C in a 500-ml., ice-cooled three-neck flask provided with a stirrer and a thermometer and placed under a good hood. Then a total of 127 g. (0.5 mole) of iodine is added in portions of 3-4 g. with vigorous stirring. Each new portion is added only after the previous one has completely reacted. The CNI product is extracted 10 minutes after the end of the addition, first with 120 ml., then with 100 ml. and finally with 80 ml. of ether. The combined ether extracts are concentrated in vacuum at room temperature. This yields 90 g. of impure, light-brown product. To remove the NaI, which is soluble in the ether solution of CNI  $[NaI_2(CN)]$ , the crude product is heated to 50°C with 120 ml. of water and shaken for 15 minutes at slightly reduced pressure (about 0.5 atm.). After cooling at 0°C, the colorless, crystalline CNI is separated from the yellow mother liquor by filtration, washed repeatedly with small amounts of ice water, and dried in air (under a hood). The yield is 59 g. (77% based on I).

Cyanogen iodide of highest purity is obtained by recrystallization from chloroform. The total product is dissolved in 150 ml. of boiling chloroform and slowly cooled to  $-10^{\circ}$ C. After filtering, it is again washed with some ice-cold chloroform and dried in air.

II. A modification of this process, in which half of the  $I_2$  is not wasted to form alkali iodide, has been described by Grignard and by Zmaczyński. In accordance with the equation

$$2 \text{ KCN} + \text{I}_2 + \text{Cl}_2 = 2 \text{ CNI} + 2 \text{ KCl}$$

 $Cl_2$  is introduced while the KCN is reacting with the  $I_2$ , or the stoichiometric quantities of  $H_2SO_4$  and  $I_2$  are added to a KCN (NaCN) solution, i.e., one mole of  $I_2$  per mole of starting cyanide. At the end of the reaction, the same amount of KCN or NaCN is added and  $Cl_2$  is introduced until one drop of the solution turns starchiodide paper blue. The temperature is held at 0 to  $+5^{\circ}C$  during the entire process. Workup is the same as described above.

PROPERTIES:

Colorless, silky crystalline needles. M.p. (in sealed tube) 146.5°C. Sublimes. Slightly soluble in cold, and readily soluble in hot water; soluble in alcohol and ether.

**REFERENCES:** 

- I. B. Bock and A. Hillebert. Organic Syntheses <u>32</u>, 29, New York-London, 1952.
- II. V. Grignard and P. Crouzier. Bull. Soc. Chim. France [4] 29, 215 (1921); E. Zmaczyński. Ber. dtsch. chem. Ges <u>59</u>, 711, (1926).

## Cyanic Acid

## HNCO

Heating of urea yields cyanuric acid, which is converted to cyanic acid by dry distillation.

## A) CYANURIC ACID (CONH)<sub>3</sub>

Urea is heated and the resulting crude product is twice recrystallized from hot water. After the first filtration, 10 ml. of concentrated hydrochloric acid is added to one liter of the solution.

Commercial cyanuric acid is purified by recrystallization in the same manner. This step must not be omitted, as otherwise the cyanuric acid will explode on distillation at  $-30^{\circ}$ C, the explosion being accompanied by polymerization.

## **B) CYANIC ACID HNCO**

A 1-m.-long, 25-mm. I.D. Vycor tube is filled with dehydrated cyanuric acid for a length of about 700 mm. in such fashion that a narrow channel remains for dry N<sub>2</sub>, admitted at one end of the tube. The other end of the tube is connected to a 200-ml., two-neck receiver immersed in a cooling bath (Dry Ice-ether). The flow of N<sub>2</sub> is started and the empty section of the tube is brought to red heat with a 250-mm.-long tubular furnace. When the empty section is hot, the end of the furnace is moved toward the cyanuric-acid-filled section and the furnace is advanced as decomposition of the acid proceeds. The reaction product, condensed in the receiver, is evacuated for several hours at  $-80^{\circ}C$  (using an oil pump), shaken with P<sub>2</sub>O<sub>5</sub> at  $-20^{\circ}C$ , and distilled into a receiver cooled to  $-80^{\circ}C$ . The final purification proceeds in high vacuum.

The connection between the reaction tube and the receiver must be of large diameter to prevent plugging. The sublimate appearing in the tube itself is not heated, as it will form a great deal of hard-to-separate HCN.

Removal of HCN. Prior to the  $P_2O_5$  treatment, the reaction product is shaken for several hours with some  $Ag_2O$ , and distilled over  $P_2O_5$ . The operation is repeated if necessary.

The yield is 80%.

Determination of the HCN content of HNCO. A thin jet of 1 ml. of HNCO is added from a precooled pipette to 40 ml. of  $0^{\circ}$ C, 1N, vigorously shaken KOH solution. The HNCO is absorbed as cyanate. The CN<sup>-</sup> ion is titrated with silver nitrate solution by the Liebig method.

#### PROPERTIES:

Formula weight 43.03. Colorless liquid with stifling odor. B.p.  $23.5^{\circ}$ C; vapor pressure(0°C) 271.0 mm. Soluble in water (dec.). Polymerizes below 150°C to a trimer (HNCO)<sub>3</sub> and converts above 150°C to cyanuric acid. At 0°C, liquid cyanuric acid polymerizes within an hour to a mixture of the two substances. Dilute solutions of cyanic acid in ether, benzene or toluene are stable for weeks. d (liq., -20°C) 1.156. Heat of formation -36.5 kcal./mole. REFERENCE:

M. Linhard. Z. anorg. allg. Chem. 236, 200 (1938).

# Hydrogen Thiocyanate

## HNCS

 $\frac{\text{KNCS} + \text{KHSO}_4 = \text{K}_2\text{SO}_4 + \text{HNCS}}{97.2 \quad 136.2 \quad 174.3 \quad 59.1}$ 

A mixture of KNCS and KHSO<sub>4</sub> is ground together as finely as possible and allowed to stand for 3-4 weeks over  $P_2O_5$ . Then flask *a* of an apparatus such as that shown in Fig. 216 is successively charged (the order of addition should be maintained) with 250 g. of glass beads (6-7 mm. in diameter), 100 g. of KNCS alternated with beads, then a covering layer of glass beads, then the required amount of KHSO<sub>4</sub> mixed with glass beads. Without disturbing these layers, the glass flask is mounted in a rigid support and joined on one side of the horizontal tube *p* via a ground glass spray trap *f*. The other ground glass joint of *a* is closed off with a cap.

Condensation occurs in the two vessels c and d, in series, the necks of which end in male ground joints. A manifold with ground glass joints connects one neck of each of the vessels to a mercury manometer. An iron-Constantan thermocouple, sealed into a ground joint cap, is inserted into the other neck of each vessel. A side tube from vessel c connects it to three-way stopcock  $h_1$ , and vessel d is similarly connected to three-way stopcock  $h_2$ . The latter is connected to the pump via a drying tower filled with calcium hydroxide. Air may be admitted via  $h_2$  as required.

Before starting the reaction the vessels are thoroughly dried by heating in vacuum, and then vessel c is cooled, with the vacuum on, in a liquid nitrogen bath. The stopcock leading to drying tube p is opened. Now flask a is rotated on its axis for 5 to 10 minutes, causing the salts to mix and to be ground by the glass beads. The mixing causes HNCS to be generated, cooling flask a. It is held at room temperature by a lukewarm water bath. The mixture becomes rose-colored at first  $[Fe(NCS)_3]$  and then turns white to ivory. This color should be maintained until the reaction ends. If the mass becomes yellow, it means that HNCS is decomposing, yielding HCN. In this case, stopcock  $h_1$  must be closed and all further gaseous product is discarded. If the preparation is properly performed, the ivory color remains for 1-1.5 hours. If repeated rotation of a results in no further generation of HNCS, the spray device b is actuated, which drenches the mass with water vapor



Fig. 216. Preparation of hydrogen thiocyanate. a) Pyrex gas generator, two liters; b) spray device with two capillary nozzles and stopcock;  $\mathcal{I}$ ) spray trap;  $\mathcal{P}$ ) drying tube, 500 mm. long, 50 mm. I.D., filled with  $P_2O_5$ , quartz wool, and short glass rods;  $\mathcal{Q}$ ) fritted glass filter;  $h_1$ ) three-way stopcock, 20 mm. I.D.; c and d) condenser traps, 400 ml.;  $t_1$  and  $t_2$ ) iron-Constantan thermocouples;  $h_2$ ) three-way stopcock; m) Hg manometer; k) analysis tubes.

and finely divided water droplets when the stopcock is opened and flask a is vigorously rotated. Inasmuch as the addition of water significantly influences the amount and purity of the HNCS, it must be held within close limits. The control of the spray is learned with some practice. As the condensation proceeds, the Dewar flask at c is moved higher and higher, until finally the entire enlarged section of flask c is cooled. If obstructions to gas flow occur in drying tube p, then the tube is rotated a few times. The run is ended after 1 to 1.5 hours by closing stopcock  $h_1$ . It is terminated earlier if the mass in the flasks becomes yellow or the  $P_2O_5$  in tube p is exhausted or becomes bright yellow at the points where an obstruction to flow is produced (decomposition of concentrated HNCS solutions in the presence of mineral acids). Finally the product is distilled (in high vacuum) from c into d. The preparation is virtually pure (m.p. checked with thermocouple  $t_2$ ). The yield is 15-20 g. of solid HNCS.

Analysis of the solid condensate. One part of the substance is distilled into tube k, constricted for sealing off (Fig. 216). This tube

is then fused and removed from the apparatus, placed in a measured quantity of excess 0.1N KOH, and the tip is broken off. Removal of the glass fragments by filtration and weighing gives the weight of material by difference A 100-ml. portion of the filtrate is diluted to 250 ml. and the excess KOH is back-titrated with 0.1N HCl. Another 100 ml. of the filtrate is acidified with nitric acid, and the HNCS content is determined by the Volhard method.

Test for HCN. Volhard method: Titration with  $AgNO_3$  solution in the presence of KI [L. Birckenbach and K. Sennewald. Liebigs Ann. Chem. 512, 38 (1934)].

Other preparative possibility: Preparation of a dilute aqueous solution of HNCS from  $NH_4NCS$  and water in presence of the hydrogen form of an ion-exchange resin [R. Klement, Z. anorg. allg. Chem. <u>260</u>, 268 (1949)].

SYNONYMS:

Thiocyanic acid, sulfocyanic acid.

PROPERTIES:

White substance with an enamel luster. M.p.  $-110^{\circ}$ C.

Polymerizes between -90 and  $-85^{\circ}$ C to a white crystalline mass. Converts with careful heating in vacuum to bright-yellow, ether-soluble thiocyanuric acid [(HNCS)<sub>3</sub>?], which readily dissocates into HNCS. If either HNCS or the yellow form is heated at room temperature without evacuation, the material slowly turns dark red, both in the presence and in the absence of air. At about  $+3^{\circ}$ C, a rapid exothermic reaction occurs. The entire mass foams and coalesces into a slurry.

Miscible with water in all proportions. Dilute solutions (<5%) are stable. Very strong acid.

**REFERENCE:** 

L. Birckenbach and E. Buchner. Ber. dtsch. chem. Ges. 73, 1153 (1940).

## Thiocyanogen

(SCN)<sub>2</sub>

I. A suspension of AgNCS in  $CS_2$  is treated with  $Br_2$ . The AgBr is split off, releasing (SCN)<sub>2</sub>.

A) SILVER THIOCYANATE AGNCS

 $AgNO_3 + KNCS = AgNCS + KNO_3$ 169.9 97.2 166.0 101.1 A solution of  $AgNO_3$  is precipitated with the stoichiometric quantity of KNCS; the precipitate is washed by decantation, filtered and dried at  $70^{\circ}C$ .

B) THIOCYANOGEN (SCN)<sub>2</sub>

 $\begin{array}{rrrr} 2 \text{ AgNCS } + \text{ Br}_2 = 2 \text{ AgBr} + (\text{SCN})_2 \\ 331.9 & 159.8 & 375.6 & 116.2 \end{array}$ 

A suspension of 17 g. of AgNCS in 50 ml. of  $CS_2$  (distilled over  $P_2O_5$ ) is mixed with 8 g. of  $Br_2$ . The suspension is swirled around in the flask during addition. The reaction ends after a few minutes. The resultant AgBr (plus some AgNCS) is filtered off and the solution is cooled in a specimen tube to about  $-70^{\circ}C$  (Dry Ice-acetone cooling bath), precipitating (SCN)<sub>2</sub>. The thiocyanogen is separated from the  $CS_2$  in the apparatus illustrated in Fig. 217, almost all

of which is submerged in a cooling bath held at  $-70^{\circ}$ C (the fritted glass filter should be about 2 cm. below the surface). When filtration is complete, the filter tube is immediately closed off with a rubber stopper and the apparatus is evacuated. After a while, dry air is introduced, the apparatus is removed from the cooling bath, and the receiver containing the mother liquor is rapidly replaced by an empty one. Some concentrated  $H_2SO_4$  is placed in the latter to absorb moisture. The apparatus is then replaced in the cooling bath and evacuated, and once again dry air is introduced after a period of standing in the bath. The operation is repeated three or four times. The yield is 4-5 g. of (SCN)<sub>2</sub> (70-90%).

II. Very pure thiocyanogen is obtained by dissociation of nitrosyl thiocyanate in vacuum.

NOCl + KNCS = ONSCN + KCl  $65.5 \quad 97.2$   $2 ONSCN = 2 NO + (SCN)_2$ 

Fig. 217. Preparation of solid thiocyanogen. *I* filter tube; *U*) receiver (about 50 ml.); *r*) capillary connected to the pump via stopcock h.

The starting materials are NOCl, free of chlorine and nitric oxide, 2 to 4g. of which is sealed into ampoules with break-off ends, and anhydrous KNCS, somewhat less than stoichiometrically required for reaction with the NOCl (1 g. of NOCl is equivalent to 1.484 g. of KNCS). About 25 ml. of liquid SO<sub>2</sub> per g. of NOCl (the SO<sub>2</sub> is dried over  $P_2O_5$ ) is used as the solvent. Finally, a high-vacuum apparatus must be available, to which the



Fig. 218. Preparation of thiocyanogen. dumbbell apparatus of Fig. 218 and a Stock tip breaker (Fig. 35) can be attached. The latter is used for opening the nitrosyl chloride ampoules.

The KNCS is introduced into the bottom sphere I, and enough  $SO_2$  is condensed on it to cover the salt with a 1.5-cm. layer of liquid. The suspension is then frozen in a liquid-nitrogen bath and the NOC1 and finally the remaining  $SO_2$  are condensed on top of the frozen layer. The reaction begins immediately after thawing. The apparatus must be occasionally agitated. The reaction temperature should be about  $-30^{\circ}C$ .

After one hour the resulting deep-red suspension is cooled to  $-50^{\circ}$ C. The apparatus must be immersed in the cooling bath to above the level of the fritted glass filter. The product is evaporated by cooling the flask that originally contained the solvent in a liquid-nitrogen bath. The NOSCN product decomposes at the same time, liberating NO, which is also frozen by the liquid-nitrogen bath. The evaporation must be performed slowly and with care and requires about 10 hours. After the red NOSCN color has almost entirely disappeared, the same amount of solvent is again condensed on the colorless (SCN)<sub>2</sub> and the solution is filtered into sphere II. After another careful evaporation at  $-30^{\circ}$ C, completely colorless, pure (SCN)<sub>2</sub> is obtained in 100% yield.

Additional preparative method: Electrolysis of  $NH_4NCS$  in methyl alcohol solution [H. Kerstein and R. Hoffmann. Ber. dtsch. chem. Ges. 57, 491 (1924)].

## PROPERTIES:

White to pale yellow crystals. M.p.  $15-16^{\circ}$ C. At room temperature, explosive conversion to a brick-red, solid material. Decomposed by water to HCN, HNCS and  $H_2SO_4$ . When SCN<sup>-</sup> is added to an aqueous (SCN)<sub>2</sub> solution, pale yellow (SCN)<sub>3</sub> is formed. More electronegative than  $I_2$ ; liberates  $I_2$  from iodides. Soluble in alcohol, ether, CS<sub>2</sub> and CCl<sub>4</sub>.

#### **REFERENCES:**

- I. E. Söderbäck. Liebigs Ann. Chem. <u>419</u>, 217 (1919).
- II. F. Seel and D. Wesemann. Chem. Ber. 86, 1107 (1953).

## Ammonium Trithiocarbonate

 $(NH_4)_2CS_3$ 

Ammonium pentasulfide (see p. 369) is digested with  $CS_2$  in a wide-neck flask equipped with a reflux condenser and an immersed, water-cooled coil. Colorless  $(NH_4)_2S$  precipitates on the cold surface of the cooling tube, but is later converted to the pale orange-yellow thiocarbonate.

Other preparative methods: A pure product is obtained from liquid  $NH_3$ ,  $H_2S$  and  $CS_2$ .

PROPERTIES:

Formula weight 144.27. Pale orange-yellow, very hygroscopic crystals. Decomposes forming  $CS_2$  and  $(NH_4)_2S$ , or  $CS_2$ ,  $NH_3$  and  $NH_4HS$ . After a long time, the thiocyanate is also formed.

Very soluble in water; slightly soluble in alcohol and ether, soluble in liquid  $NH_3$ , giving a dark-red color. Red, oily  $H_2CS_3$  is produced on addition of concentrated hydrochloric or sulfuric acids.

REFERENCES:

H. Mills and P. L. Robinson. J. Chem. Soc. (London) <u>1928</u>, 2330;
G. Gattow, unpublished.

## **Barium Trithiocarbonate**

#### BaCS<sub>3</sub>

Prepared by the action of  $CS_2$  on  $Ba(HS)_2$  solution.

Barium hydroxide is dissolved in  $CO_2$ -free distilled water and any precipitated  $BaCO_3$  is filtered off in a nitrogen atmosphere. The solution is then saturated with  $H_2S$ . A  $CS_2$ -saturated stream of  $N_2$  is then bubbled through the solution. After addition of ether, the red solution precipitates yellow  $BaCS_3$ . It is separated from the aqueous phase and dried in vacuum or in a stream of  $N_2$ .

Other preparative methods: I. Carbon dioxide-free Ba(OH)<sub>2</sub> is suspended in alcohol, the air is displaced with N<sub>2</sub>, and a CS<sub>2</sub>saturated stream of N<sub>2</sub> is bubbled through the solution. The Ba(OH)<sub>2</sub> disappears and a yellow precipitate is produced. This is further treated as described above.

II. According to R. Klement and W. Schmidt [Naturwiss. 42, 154 (1955)] BaCS<sub>3</sub> can also be obtained on an ion exchanger. A strongly basic OH<sup>-</sup> form anion exchanger is converted to the S form by means

of aqueous  $H_2$  S or Na<sub>2</sub> S. The ion exchanger acquires a greenish color. After washing with water, CS<sub>2</sub> is added to the aqueous suspension of exchanger in an amount corresponding to the exchange capacity. The mixture is shaken until no CS<sub>2</sub> droplets can be discerned. Thus, for example 50 g. of air-dried Amberlite IRA 410 (exchange capacity 4.2 meq./g.) and 4g. of CS<sub>2</sub> may be used. The ion exchanger, which is now salmon-colored, is placed in an ordinary glass column and eluted with 1.5N BaCl<sub>2</sub> until the yellow color of the eluate disappears. When ethanol is added to the eluate, 16 g. of analytically pure BaCS<sub>3</sub> (62% of theoretical) is obtained. The exchanger may be reused after successive treatment with acid (2N HCl, for example), base and sulfide.

## PROPERTIES:

Formula weight 245.57. Yellow, microcrystalline powder or yellow, hexagonal double pyramids; stable in air; dissolves in water, giving a red color. On heating, decomposes to BaS and  $CS_2$ .

Solubility: 1.08 g./100 g. of  $H_2O(0^{\circ}C)$ , 1.5 g./100 g. of  $H_2O(20^{\circ}C)$ . An aqueous solution of BaCS<sub>3</sub> dissolves one atom of S per molecule of trithiocarbonate. Heat of formation -130.1 kcal./mole.

**REFERENCES:** 

E. W. Yeoman. J. Chem. Soc. (London) 119, 38 (1921).

G. Gattow. Symposium über Thermodynamik in Fritzen-Watten (Symposium on Thermodynamics at Fritzen-Watten, Austria), August 20-25, 1959, 19 (1-3), 1959.

## SECTION 12

## Silicon and Germanium

## P. W. SCHENK

## Silicon

#### Si

The starting material for preparation of high-purity silicon is either commercial silicon, purified by recrystallization from molten aluminum, or silicon produced by Von Wartenberg's modification of Kühne's method, explained below.

The reaction vessel is a Hessian ceramic crucible or a watercooled copper ingot mold, which is first packed with finely powdered corundum in the following manner: The bottom is filled to a height of about 1 cm. with corundum powder (sieved through a 0.15-mm. screen), an aluminum tube with a diameter of about 8 cm. (with as smoothly polished a surface as possible) is inserted vertically, and the space between the A1 tube and the crucible wall is tightly packed with powdered corundum. The reaction mixture, consisting of one part of A1 pellets (99.995%), one part of very pure sulfur (purified by the Bacon and Fanelli or by the Von Wartenberg and one part of the purest available p. method. see 342). quartz sand (rock crystal powder such as used for making transparent fused quartz items—it is especially low in boron), is then packed inside the tube to a height of about 5 cm. The Al tube is then pulled upward 4 cm. and the process is repeated. After three repetitions, the crucible is full. It should not be filled all at once because the friction would make it impossible to withdraw the Al tube. Finally, another 0.5-cm. layer of Al pellets is added and the mass is ignited with an H<sub>2</sub>-O<sub>2</sub> torch in order to preclude any contamination of the charge. After cooling, the slag and the product block are easily dumped out; the hard white slag is readily broken up with a hammer. The block with the product contained in it is broken down with water, the Al(OH), is dissolved with hydrochloric acid, and the alumina (corundum) thus obtained is filtered off and washed. It may then be reused. The Si block is extracted for several days with boiling 1:2 hydrochloric acid, HF-H<sub>2</sub>SO<sub>4</sub>. The residue is then melted washed and boiled with

for half an hour with  $\rm KHSO_4$  and is then thoroughly washed. From each 350 g. of Al-SiO<sub>2</sub> charge, 120 g. of crude product is obtained. This yields 75 g. of pure silicon, which is then smelted in a stream of argon and chlorine (as shown below) to remove Al and other metal traces. The Si recrystallized from molten Al may also be purified by the following procedure:

A quartz crucible 12 cm. long and 2 cm. in diameter (a pure quartz tube fused at one end) is filled with 20 to 30 g. of powdered Si. This crucible is inserted into another fused quartz tube, which is 60 cm. long and 3 cm. in diameter, and has a male ground joint at the top (Fig. 219). The 5-mm, argon inlet tube is fused at the



Fig. 219. Smelting of silicon under chlorine according to Von Wartenberg.

bottom to a 2-cm.-diameter outer tube. which in turn is sealed to the top of a cap equipped with a female ground joint. The argon thus returns through the annulus between the 2-cm, tube and the quartz tube and may therefore be recovered for reuse. The exit gases flow to the hood through a bubble counter filled with  $H_2SO_4$ . The quartz tube is slipped into a 40-cm.-long platinumrhodium-ribbon furnace; a shielded thermocouple is inserted immediately underneath the tube. The furnace is brought to a temperature of 1470°C, as measured at the bottom of the reaction tube. An oxygen-free stream of argon (purified in the same manner as N<sub>2</sub>, see p. 458) is fed to the inlet tube via a feed line constructed exclusively of ground-joint-connected elements. Some pure chlorine may be mixed with argon by means of a tee joint. First, all air is displaced from the apparatus until no oxygen can be detected at the outlet (see p. 336, section on oxygen). Heating is then begun and, when the Si is melted, the chlorine stream is started at a rate which will allow 2 to 3 liters of Cl<sub>2</sub> to flow through the apparatus in the course

of about 3 hours. Besides  $SiCl_4$ , all impurities except boron are thus volatilized. The chlorine flow is then shut off, the temperature allowed to drop to  $1300^{\circ}C$ , and the furnace removed. The reaction tube may be removed from the stand after about half an hour, and the crucible (which is always cracked) may then be dumped out. The product block is left in pure concentrated hydrofluoric acid for 24 hours so that all the adhering quartz particles may be removed. Silicon prepared in this manner has a resistivity of 15-17 ohm.

Further purification is possible by the zone melting technique of Pfann. In order to avoid contamination from the walls of the crucible during the melting, a technique similar to that of Verneuil-Miethes is recommended.

Other preparative methods:

This is basically an old process. As used by DuPont, it is arranged as follows:

The quartz glass tube A (Fig. 220), 180 cm. long and 20 cm. in diameter, is heated along its entire length by an electric furnace.



Fig. 220. Preparation of silicon by the DuPont method.

Zinc vapor, produced from high-purity zinc, flows in through B; preheated SiCl<sub>4</sub> vapor enters at C through the coil D. The apparatus is preheated for 5 hours before the reagent vapors are allowed to enter. The two baffles,  $K_1$  and  $K_2$ , are pierced by openings (5-cm. diameter) and break up the rather fast flow of reagent vapors. The yield is about 3 to 4 kg. of silicon over a period of about 24 hours. The product is supposed to be spectroscopically pure after washing with water. This pilot plant apparatus has been miniaturized for laboratory purposes by Von Wartenberg. Further purification is accomplished by his method of melting under Cl<sub>2</sub>, given above.

## PROPERTIES:

Friable, mirrorlike (when polished) substance with a bluish luster. Insoluble in acids, including HF. Soluble in sodium hydroxide, evolving H<sub>2</sub>. M.p.  $1423^{\circ}$ C, b.p.  $2630^{\circ}$ C; d 2.4. Lamellae or octahedral crystals. Diamond structure.

## **REFERENCES:**

A. Kühne, Z. anorg. allg. Chem. <u>99</u>, 123 (1917); German Patent 147 871 Cl. 13 i, Chem. Zentr. <u>1904</u> I, 64; D. W. Lyon, J. Electrochem. Soc. <u>96</u>, 359 (1949); H. von Wartenberg, Z. anorg. allg. Chem. 265, 186 (1951); 283, 372 (1956); 286, 247 (1957); F. B. Litton and H. C. Anderson, J. Electrochem. Soc. 101, 287 (1954); H. C. Theuerer, Bell Lab. Rec. 33, 327 (1955), Chem. Zentr. 1956, 9938. For literature on zone melting, see W. G. Pfann, Chem. Eng. News 34, 1440 (1956); H. Kleinknecht, Naturwiss. 39, 400 (1952); S. Müller, Z. Naturforschg. 9b, 504 (1954); F. Trendelenburg, Angew. Chem. 66, 520 (1954). For melting without crucibles, see P. H. Keck, S. B. Levin, J. Broder and R. Liebermann, Phys. Rev. (2) 92, 847; Bull. Amer. Phys. Soc. 28, 11 (1953), Chem. Zentr. 1954, 7362; C. P. Kempter and C. Alvarez-Tostado, Z. anorg. allg. Chem. 290, 238 (1957).

#### Silanes

#### SiH<sub>4</sub> (Si<sub>2</sub>H<sub>6</sub>, Si<sub>3</sub>H<sub>8</sub>)

Silicon hydrides are prepared either by acid decomposition of magnesium silicide or by reduction of SiCl<sub>4</sub> with LiAlH<sub>4</sub>. Only a relatively modest yield of silane (20-30%) is obtained by the aqueous acid decomposition of silicide, but the proportion of the higher silanes is somewhat greater. If the reaction is carried out in liquid NH<sub>3</sub> or N<sub>2</sub>H<sub>4</sub> instead of water, appreciably higher yields (over 80%) are obtained; besides, up to 90% of the product consists of SiH<sub>4</sub>, if one neglects the easily separable H<sub>2</sub>.

The reaction vessel (Fig. 221) is filled to about two thirds of its height with dry liquid  $NH_3$  (see p. 460). Enough  $NH_4Br$  is dissolved in the  $NH_3$  so that an approximately 50% excess is



Fig. 221. Preparation of silane.

present; i.e., 85-90 g.  $NH_4Br$  should be used for each 10 g. of  $Mg_2Si$ . The finely divided silicide (see p. 921) is placed in the side flask b. A slow stream of  $NH_3$  is introduced through tube c and vilowed to bubble through the liquid to provide agitation; portions of the silicide are slowly added by rotating the small flask in the joint g. The gas evolved flows through the collecting tube e, which is filled with water in order to remove the excess  $NH_3$ . The well-boiled water, acidified with HCl in order to prevent decomposition of the silane, may be replenished as required with the aid of the leveling bulbs i and j. From the collecting tube, the gas flows through tube k, which is packed with  $P_2O_5$  to remove the water, and is then condensed in a trap l, which is cooled with liquid  $N_2$ . It is finally fractionated by Stock's procedure (see Part I, p. 66), using high vacuum equipment.

 $LiAlH_4 + SiCl_4 = SiH_4 + LiCl + AlCl_3$ 38.0 169.9 32.1 42.4 133.4

A solution of 0.348 g. of  $LiAlH_4$  (9.19 mmoles) in 12.5 g. of peroxide-free ether is placed in a reaction vessel, which is attached to a vacuum pump through a ground glass joint. The solution is then frozen and 1.33 g. of  $SiCl_4$  (7.82 mmoles) is condensed on the solidified solution (the  $SiCl_4$  is best measured out in the gaseous state). The reactor is now permitted to warm slowly. At about 0°C, a vigorous gas evolution commences. The products are condensed in a liquid-nitrogen-cooled trap, then distilled at -159°C into a trap, also cooled with liquid N<sub>2</sub>. The yield is 175 ml. (99%) of pure SiH<sub>4</sub>.

When LiH (even in very large excess) is used instead of LiAlH<sub>4</sub>, the reaction gives poorer yields. See Peake, Nebergall and Yun Ti Chen for an apparatus used in preparation of larger quantities of SiH<sub>4</sub> by that method.

The starting material for preparation of  $\text{Si}_2\text{H}_6$  is  $\text{Si}_2\text{Cl}_6$ . An LiAlH<sub>4</sub> solution is added slowly, drop by drop, to  $\text{Si}_2\text{Cl}_6$  in ether solution. The LiAlH<sub>4</sub> excess should be about 15%. The  $\text{Si}_2\text{H}_6$  yield is about 8%.

PROPERTIES:

SiH<sub>4</sub>: M.p. -185°C, b.p. -111.9°C; d (at m.p.) 0.68. Si<sub>2</sub>H<sub>6</sub>: M.p. -132.5°C, b.p. -14.5°C; d (-25) 0.685. Very reactive. Ignites in air.

## REFERENCES:

A. Stock, Hydrides of Boron and Silicon (Ithaca, 1933). Additional references to the literature are found there. A. Stock, Z. Electrochem. <u>32</u>, 341 (1926); W. C. Johnson and S. Isenberg, J. Amer. Chem. Soc. <u>57</u>, 1349 (1935); A. E. Finholt, A. C. Bond, K. E. Wilzbach and H. J. Schlesinger, J. Amer. Chem. Soc. <u>69</u>, 2692 (1947); F. Fehrér and W. Tromm, Z. anorg. allg. Chem. <u>282</u>, 29 (1955); J. S. Peake, W. H. Nebergall and Yun Ti Chen, J. Amer. Chem. Soc. <u>74</u>, 1526 (1952); H. S. Gutowsky and E. O. Stejskal, J. Chem. Physics, <u>22</u>, 939 (1954).

## Polysilanes

$$(SiH)_x$$
,  $(SiH_2)_x$ 

$$2 \operatorname{SiHBr}_{3} + 3 \operatorname{Mg} = 2 (\operatorname{SiH}) + 3 \operatorname{MgBr}_{2} \\ 537.7 \quad 73.0 \quad 58.2 \quad 552.5$$

Magnesium turnings (6 to 8 g.) are placed in the reaction chamber a (Fig. 222), which is separated by a perforated plate b from the filter g; the reaction chamber is evacuated and filled with pure N<sub>2</sub> (see p. 458). Vessel c is filled with a 30 vol. % solution of



Fig. 222. Preparation of (SiH)<sub>x</sub>

SiHBr, in absolute ether. With stopcocks e and f still closed, stopcock dis opened to admit just enough SiHBr<sub>3</sub> solution to cover the Mg turnings. The heat of the instantly commencing reaction causes the ether to boil. The Mg is covered with a yellow incrustation which is broken up by the action of the hand-operated agitator h. The reaction product collects on the filter q. After about 20 to 30 minutes, the ether and the magnesium bromide etherate are drawn by suction into iand are then released into k. Absolute ether from vessel c is now let in via stopcock d and the apparatus is washed several times until test samples drawn out through the two-way stopcock l are free of Br. The reaction product is now suspended in ether and, by tipping the apparatus, it is made to overflow into flask *m*; the ether is now pumped out to obtain the dry product. Yield is about one gram.

The apparatus is similar to that illustrated in Fig. 222, except that stopcocks e and f are omitted along with the reflux condenser

and the perforated plate b. A 5-g. quantity of CaSi (see p. 946 for preparation) is poured from flask m (by rotating the flask in the joint) into 300 ml. of HCl-saturated absolute alcohol, which covers the glass filter plate g. The procedure must be carried out in an atmosphere of dry CO<sub>2</sub> with good agitation. The addition of CaSi takes about an hour. Good cooling must be provided. After standing overnight while the evolution of H<sub>2</sub> ceases, the brown precipitate is separated in a CO<sub>2</sub> atmosphere from the supernatant liquid. The precipitate is first washed with ice-cold absolute alcohol and then with absolute ether.

The preparation of the higher unsaturated Si hydrides may be accomplished by cleaving the saturated hydrides with electrical discharge.

#### PROPERTIES:

 $(SiH)_X$ : lemon-yellow substance. Hard, amorphous; oxidizes slowly in air.

 $(SiH_2)_X$ : brownish substance. The dry material ignites spontaneously in air, leaving a SiO<sub>2</sub> residue which may be gray because of the presence of iron silicide. Evolves H<sub>2</sub> with alkali hydroxides.

#### REFERENCES:

G. Schott and W. Herrmann, Z. anorg. allg. Chem. 288, 1 (1956); G. Schott and E. Hirschmann, Z. anorg. allg. Chem. 288, 9 (1956); A. Stock and K. Somieski, Ber. dtsch. chem. Ges. 54, 524 (1921); 56, 247 (1923); R. Schwarz and F. Heinrich, Z. anorg. allg. Chem. 221, 277 (1935).

## Silicon Tetrachloride

SiCl₄

 $Si + 2Cl_2 = SiCl_4$ 28.1 141.8 169.9

Silicon (prepared as shown above) or coarsely ground ferrosilicon (which should contain as much Si as possible) is placed in a boat inserted into a Pyrex tube about 60 cm. long and 2-3 cm. in diameter, through which a stream of  $Cl_2$  is passed (Fig. 223). The  $Cl_2$  is predried over concentrated  $H_2SO_4$ . A condenser is attached to the other end of the tube by means of an adapter; the tube itself is heated in an electric furnace. The tube should incline toward the condenser to prevent the SiCl<sub>4</sub> from backing up.



Fig. 223. Preparation of silicon tetrachloride.

The condenser end of the reaction tube should project out of the furnace for some distance so that the invariable byproduct FeCl<sub>3</sub> may deposit without plugging the condenser. The condenser discharges into a distilling flask which is set in an ice bath. All joints must be tight and the entire apparatus carefully dried before the start of the run. A CaCl<sub>2</sub> tube is attached at the end of the side arm of the distilling flask. If this precaution is overlooked, the side arm will immediately be plugged with silicic acid produced by reaction with atmospheric moisture. The reaction tube is heated to about 400°C and the Cl<sub>2</sub> flow is then started.

If, in addition to  $SiCl_4$ , the higher Si chlorides are desired  $(Si_2Cl_6 \text{ and } Si_3Cl_8)$ , the temperature of the tube should be as low as possible, but no lower than just below 400°C. When the reaction is well established, heating may be sharply reduced since the reaction itself evolves considerable heat. The crude chloride, which is accumulated in the distilling flask, may be purified by fractional distillation. If an absolutely Cl-free preparation is desired, a second distillation over Cu turnings is performed. Both distillations must be run in absolutely dry equipment. The ampoules into which the SiCl\_4 is distilled should be fused to the distillation apparatus, since it is impossible to obtain a non-turbid product if this is not done. Yield is quantitative.

#### PROPERTIES:

Clear, colorless liquid; fumes heavily upon exposure to air. Rapidly hydrolyzes in water to form a  $SiO_2$  gel. Miscible with benzene, ether, chloroform and saturated hydrocarbons. Forms esters of silicic acid with alcohols. B.p. 57.5°C, m.p. -68°C; d 1.52.

## **Higher Silicon Chlorides**

If the preparation of  $SiCl_4$  is carried out at temperatures below  $400^{\circ}C$ , it is possible to isolate very small quantities of  $Si_2Cl_6$  and  $Si_3Cl_8$  from the residue of the final distillation.

Chlorides up to SigCl<sub>14</sub> are best prepared by chlorination of calcium silicide. The procedure is as follows: Cl<sub>2</sub>, dried over concentrated H<sub>2</sub>SO<sub>4</sub>, is passed through a vertical glass tube (34-mm. diameter, about one meter long), half filled with bean-size lumps of Ca silicide (about 30-35% Ca). The tube should not be too full, for then it might plug during the run; about 200-250 g. is used. The tube is placed inside a short, movable electric heating coil. It is important that the reaction take place at the lowest possible temperature and that only a short section of the tube be heated at any time. The reaction starts at 250°C. The temperature must then be immediately lowered to 150°C. The Cl, flow should not exceed 100 bubbles/minute. The reaction products are accumulated in a cooled receiver via an attached condenser. After 12 to 14 days, during which the heating coil is slowly moved along the entire length of the tube, all the silicide is reacted, and about 700 ml. of chloride mixture is collected. The higher chlorides are obtained from this mixture by fractionation. If a low temperature and a slow flow rate of Cl<sub>2</sub> are used, then about 35% of the product mixture boils at a higher temperature than SiCl<sub>4</sub>. About 30% of this is Si<sub>2</sub>Cl<sub>6</sub>, 4%, Si<sub>3</sub>Cl<sub>6</sub>, and 1% represents chlorides up to Si<sub>e</sub>Cl<sub>14</sub>. After evaporation of the SiCl<sub>4</sub>, the residue is fractionated at reduced pressure. Addition of 2 to 5% of an alkali chloride, alkaline earth chloride or ammonium chloride or dilution of the chlorides (e.g., with SiCl<sub>4</sub>) should improve the yield.

#### PROPERTIES:

B.p.:  $\text{Si}_2\text{Cl}_6$ , 147°C;  $\text{Si}_3\text{Cl}_6$ , 216°C;  $\text{Si}_4\text{Cl}_{10}$ , 150°C (at 15 mm.);  $\text{Si}_5\text{Cl}_{12}$ , 190°C (15 mm.);  $\text{Si}_6\text{Cl}_{14}$  sublimes in vacuum at 200°C.

According to R. Schwarz, the higher Si chlorides, such as  $Si_{10}Cl_{22}$ , can be prepared by treatment of  $SiCl_4$  in the apparatus illustrated in Fig. 224, which the author calls a "quenching tube." The procedure is as follows.

A silicon carbide rod, held in place by two electrode connector clamps, is fitted into a Liebig condenser. The electrode clamps are sealed into Pyrex caps, which fit over the ground-glass male joints of the condenser.

If the electrical terminals of these clamps consist of watercooled copper tubing, then the seal to the glass caps is made with rubber tubing. A clamp is made by closing one end of the copper tube, welding on a piece of thin wall stainless steel tube and



Fig. 224. Preparation of the higher silicon chlorides.

splitting the end of the latter with two lengthwise saw cuts. The SiC rod fits into that end. The open end of the copper tube is closed with a cap which carries the inlet and outlet tubes for the cooling water (the inlet tube extends well inside the closed tube).

Alternatively, iron electrode clamps, sealed to the Pyrex caps with asbestos-waterglass mixture, are used with no cooling. In that case, the Pyrex caps may have to be wound with cotton string, wetted with water to remove the heat.

The "quenching tube" itself can be made of copper. In this case, the glass caps are sealed on with rubber tubing (pieces of bicycle inner tube).

The apparatus is first thoroughly dried by heating in a stream of inert gas, with the cooling water off. After brief cooling, a stream of hydrogen is introduced at a and the air is displaced. Then 25 ml. of SiCl<sub>4</sub> is placed in flask a, the hydrogen saturated with the SiCl<sub>4</sub> and the heating resumed [the hydrogen is first passed over Pd abestos to remove O<sub>2</sub> and over P<sub>2</sub>O<sub>5</sub> for drying (see section on Hydrogen)].

The first run in a new tube should be made with  $H_2$ ; Si  $_{10}Cl_{20}H_2$ is formed under these conditions. If oxygen-free argon and SiCl<sub>4</sub> are used in a later run, Si  $_{10}Cl_{22}$  is formed. The temperature of the SiC rod should be between 1000°C and 1100°C. A trap *b*, cooled with liquid nitrogen, is connected to the distilling flask which serves as the receiver.

It happens occasionally that the reaction fails to start. In that case, the gas is either not completely water-free or it contains oxygen. Whenever a new SiC rod is used, the first run should be made with  $H_2$  as the carrier gas, for otherwise the reaction will not start. From 112 g. of SiCl<sub>4</sub> the yield is about 35 g. crude product, from which the lower boiling fractions are stripped off. The Si<sub>10</sub>Cl<sub>20</sub>H<sub>2</sub> cannot be distilled without decomposing. The product prepared with argon is fractioned under high vacuum. The Si<sub>10</sub>Cl<sub>22</sub> comes over as a highly viscous oil between 215 and 220°C.

The preparation of the higher Si chlorides by passage of  $SiCl_4$  over Si at 1000°C is described by Rochow and Didtschenko; the preparation by means of a glow discharge is described by Hertwig and Wiberg.

PROPERTIES:

Viscous, with a consistency ranging from oily to honeylike. Flammable. Hydrolyzed by water.

**REFERENCES:** 

A. Besson and L. Fournier, Comptes Rendus Hebd. Séances Acad. Sci., 152, 603 (1911); C. Friedel, Comptes Rendus Hebd. Séances Acad. Sci. 73, 1011 (1871); C. Friedel and A. Ladenburg, Liebigs Ann. Chem. 203, 253 (1880); L. Gatterman and E. Ellery, Ber. dtsch. chem. Ges. 32, 1114 (1899); L. Gatterman and K. Weinlig, Ber. dtsch. chem. Ges. 27, 1943 (1894); J. W. Mellor, Comprehensive Treatise on Inorg. Chem., VI, p. 971; Int. Crit. Tabl., Vol. I, p. 162; G. Martin, J. Chem. Soc. (London) 105, 2836, 2860 (1914); Ber. dtsch. chem. Ges. 45, 2097 (1912); 46, 2442, 3289 (1913); L. Troost and P. Hautefeuille, Ann. Chim. Phys. (5) 7, 459 (1871); R. Schwarz and H. Meckbach, Z. anorg. allg. Chem. 232, 241 (1937); E. G. Rochow and R. Didtschenko, J. Amer. Chem. Soc. 74, 5545 (1952); H. S. Gutowski and E. O. Stejskal, J. Chem. Phys. 22, 939 (1954); D. F. Stedmann, U.S. Patent 2,621,111; W. J. Walton, U.S. Patent 2,602,728 (1952); H. Schaefer, Z. anorg. allg. Chem. 274, 265 (1953); C. F. Wilkins, J. Chem. Soc. (London), 1953, 3409; K. A. Hertwig and E. Wiberg, Z. Naturforsch., 6b, 336 (1951); W. C. Schumb and E. L. Gamble in H. S. Booth, Inorg. Syntheses New York and London, 1939, Vol. I, p. 42.

## Silicon Tetrabromide

SiBr<sub>4</sub>

 $Si + 2 Br_2 = SiBr_4$ 28.1 319.7 347.8

A boat, placed in a Vycor tube, is heated in an electric furnace to about  $600^{\circ}$ C. The boat contains finely powdered Si (which is best mixed with about 4% copper dust). An  $O_2$ -free stream of  $N_2$  (from a "copper tower," see p. 458), saturated with Br<sub>2</sub> vapor in a wash bottle, is passed over the silicon. A condenser with a receiver is attached to the other end of the tube, as in the preparation of SiCl<sub>4</sub>. The crude product is purified by fractional distillation.

#### PROPERTIES:

Colorless liquid. Easily hydrolyzed. Fumes in air. M.p.  $5.2^{\circ}$ C; b.p.  $152.8^{\circ}$ C; d 2.789. In contrast to SiCl<sub>4</sub> it reacts violently with metallic potassium.

**REFERENCE:** 

L. Gatterman, Ber. dtsch. chem. Ges., 22, 189 (1889).

## Silicon (II) Bromide

#### SiBr<sub>2</sub>

 $SiBr_4 + Si = 2SiBr_2$ 347.8 28.1 375.9

As shown in Fig. 225, an unglazed porcelain boat is charged with 20 g. of very pure silicon and placed in a porcelain or quartz tube (30-40 mm. diameter, 800 mm. long), placed in a Globar tubular furnace. The boat is pushed to the hottest part of the furnace. The temperature should be about  $1150^{\circ}$ C at this point. The ends of the tube, which extend beyond the furnace, are wound with lead cooling coils and are closed with well-seated rubber stoppers c and d. The stoppers are painted with quick-setting chlorinated rubber cement to ensure a better seal. The flask a, which contains SiBr<sub>4</sub>, is first cooled to at least  $-80^{\circ}$ C and is then very slowly



Fig. 225. Preparation of silicon (II) bromide.

evacuated via trap e and stopcocks b and f. Caution is necessary since the Si powder is easily entrained. When a high vacuum has been attained, the furnace is turned on. As soon as the temperature reaches 1100°C, trap e is cooled in Dry Ice or liquid nitrogen and flask  $\alpha$  is set in ice water. Crystals of Si<sub>2</sub>Br<sub>6</sub> form immediately at q and unreacted SiBr, collects in e. In six to seven hours, about 40-60 ml. SiBr, distills over, of which about 10 ml. is recovered unreacted in e. The apparatus is allowed to cool under vacuum, dry N<sub>2</sub> is introduced through f, stopper d is removed, and the tube is quickly withdrawn from the furnace. A suitable widemouth flask, which is kept ready, is pushed over the open end of the tube and, after flask  $\alpha$  is detached, a stream of dry N<sub>2</sub> is introduced via b and c into the now inclined tube. The product is melted by careful heating with an open flame (the tube is fanned with the flame) and allowed to flow into the wide-mouth flask. This operation, which takes about 15 minutes, is carried out in an atmosphere of dry N<sub>2</sub>. The product accumulated from several runs is transferred into an Anschütz distillation flask and is heated in high vacuum for half an hour at 170-180°C in order to separate the  $Si_2Br_6$ . About 80% of the product remains behind as a dark brown substance, mixed with powdered Si. This is dissolved in three to four times its volume of dry benzene and filtered while protected from moisture in an apparatus similar to that illustrated in Fig. 49, p. 71; the apparatus, however, has no heating jacket and the horizontal outlets c and d, this time provided with stopcocks, are fused instead to the middle section. The clear, dark yellow solution is freed from benzene under vacuum and is then heated for another hour in high vacuum at 170°C. Dry air is admitted after cooling and the flask containing the residue is brought to a temperature of  $-80^{\circ}$ C. At this temperature, the SiBr<sub>2</sub> is so brittle that it crumbles easily on slight shaking.

#### PROPERTIES:

Formula weight 187.92. Brown, friable substance, somewhat like shellac; transparent in thin sheets. Begins to soften at about  $60^{\circ}$ C; filaments may be drawn at  $100-110^{\circ}$ C; at  $160^{\circ}$  to  $180^{\circ}$ C, it acquires the viscosity of a heavy grease. Decomposition starts at  $200^{\circ}$ C and Si<sub>2</sub>Br<sub>e</sub> is formed. Ignites spontaneously in air at  $120^{\circ}$ C. Molecular weight from benzene, 3,000-3,600. Hydrolyzes in water, forming subsilicic acids. Strong reducing agent. Soluble in benzene, xylene, CCl<sub>4</sub>, etc.; slightly soluble in ether. Reacts explosively with strong oxidizers such as HNO<sub>3</sub>.

#### REFERENCE:

M. Schmeisser, private communication, 1958.

## Silicon Tetraiodide

Sil

 $Si + 2I_2 = SiI_4$ 28.1 507.7 535.8

The preparation is carried out in the apparatus shown in Fig. 226. It consists of a quartz tube a, which is 500 mm. long and 30 mm. in diameter and has a male ground-glass joint on both ends. Attached to the main tube at a right angle is tube b, approximately 50 mm. long, with a female ground joint at the end. The apparatus is filled in the following manner with a mixture consisting of commercial grade powdered silicon plus 4% of copper dust: a 25-cm.-long strip of copper is fashioned into a spiral which is held in place in the quartz tube by a loose plug of asbestos wool. The copper spiral is placed in such a manner that one end of it extends into the joint and the asbestos plug 1 is located about 5 to 6 cm. away from the fitting. The tube is now charged with the silicon and copper powder mixture, which is packed as loosely as possible. The tube is held vertically during the filling and the addition is continued until the copper spiral is just covered. The charge is then held in place by the loose asbestos wool plug 2. The charged tube is inserted in a tubular electric furnace, which is equipped with a thermocouple, in such a way that the upper edge of the charge is even with the end of the furnace; the apparatus is tilted about  $40^{\circ}$  from the horizontal. A receiver d with a glass outlet tube is attached at the lower joint. The upper end of tube a is closed by a fitting with a stopcock, through which a stream of oxygen-free nitrogen is introduced at a rate of four bubbles/sec. The N<sub>2</sub> is predried over concentrated  $H_2SO_4$ , silica gel and  $P_2O_5$ . Flask k, containing the iodine, is attached to the arm b. The apparatus is first thoroughly heated in a stream of N<sub>2</sub> at 600 to 700°C for one to two hours. The first addition of iodine is then made by rotating the iodine flask k in the joint; moderate heat is applied at the upper asbestos plug. Crude Sil<sub>4</sub>, contaminated with iodine, immediately flows to the lower end where it collects in the receiver. Care must be taken that solid Sil<sub>4</sub> does not plug the tube. If necessary, the Sil<sub>4</sub> must be melted down by fanning with an open flame. About 150 g. of crude  $SiI_4$  can thus be obtained over several hours. The yield, based on iodine, is 70%. Unreacted Si can be reused in later runs. For purification, the receiver is broken off at g, quickly crushed in a mortar and the crushed mass poured into a well-heated distilling flask (see Fig. 227), equipped with an approximately 20-mm.-diameter side arm. The charge a is covered with an approximately 5-mm,-thick layer of



powdered copper b and a 2-cm.-long loose plug of copper wool d is pushed up into the side arm. The distilling flask is now heated in a liquid metal bath and evacuated with an oil pump. Some SiI<sub>4</sub>, together with I<sub>2</sub>, begins to sublime at a bath temperature of 150°C; the reaction between Cu and I<sub>2</sub> begins at 180°C, and the melt begins to boil at 200°C. The stopcock on the line to the pump is now closed, and the SiI<sub>4</sub> is distilled over. Since some I<sub>2</sub> comes over at the start, the solid SiI<sub>4</sub> is rose-colored. After three distillations, pure white, solid SiI<sub>4</sub> (lemon yellow as a liquid) is obtained. Because some copper dust is occasionally entrained during the distillation, the final distillation of colorless SiI<sub>4</sub> is best done without the addition of copper. The product should be stored under an N<sub>2</sub> blanket in sealed ampoules, which are kept in a dark place.

#### PROPERTIES:

M.p.  $122^{\circ}$ C, b.p.  $287.5^{\circ}$ C. White crystalline substance which melts to a lemon-yellow liquid. Very sensitive to moisture and light; decomposes readily on heating into its elements.

According to Gomberg (see Part III, section on Adsorptive and Catalytically Active Materials),  $Si_2I_8$  is formed when  $SiI_4$  is heated together with very dry, powdered silver at 280°C for six hours in a sealed, evacuated tube.

#### REFERENCES:

R. Schwarz and A. Pflugmacher, Ber. dtsch. chem. Ges. <u>75</u>, 1062 (1941); U. Wannagat and R. Schwarz, Z. anorg. allg. Chem. <u>277</u>,

82 (1954); U. Wannagat and E. Ringel, private communication; F. B. Litton and H. C. Anderson, J. Electrochem. Soc. <u>101</u>, 287 (1954).

## Chlorosilanes

#### SiHCl<sub>3</sub>, SiH<sub>2</sub>Cl<sub>2</sub>, SiH<sub>3</sub>Cl

Several methods are available for the preparation of the three chlorosilanes. In order to obtain  $SiHCl_3$  as the principal product, the best way is to react Si with HCl in the presence of copper.

 $Si + 3 HCl = SiHCl_3 + H_2$ 28.1 109.4 135.4 22.4 l

A porcelain boat is filled with finely powdered silicon, which has been purified by boiling with hydrochloric acid and dilute hydrofluoric acid. About 10% CuCl<sub>2</sub> is mixed in with the silicon. The boat is placed in a Pyrex glass tube fitted with an adapter and a condenser, as described in the preparation of SiCl<sub>4</sub>. In this case, however, the end of the condenser should extend into the middle of the distilling flask. The apparatus is first carefully heated. Hydrogen chloride gas is produced from NaCl and H<sub>2</sub>SO<sub>4</sub>, without addition of hydrochloric acid, since it must be absolutely dry. The furnace around the tube which contains the boat is heated to 300°C and a slow stream of HCl is introduced. The receiver is cooled with acetone-Dry Ice mixture. Finally, the crude product is distilled directly from the receiver. The first fraction is HCl. The SiHCl<sub>3</sub> comes over at 36.5°C. With a careful fractionation of the forerun, SiH<sub>2</sub>Cl<sub>2</sub> may also be recovered. The yield of SiHCl<sub>3</sub> is about 50%. In order to improve the yield of SiH<sub>2</sub>Cl<sub>2</sub>, it is recommended that hydrogen be added to the HCl in the ratio of  $H_2:HCl = 4:1.$ 

Other preparative possibilities:

 $SiH_4 + HCl = SiH_3Cl + H_2 (A. Stock)$ SiCl<sub>4</sub> + H<sub>2</sub> = SiH<sub>3</sub>Cl + HCl (D. T. H u r d)

PROPERTIES:

Water-clear, mobile, flammable liquids. Very volatile. Hydrolyzed by water. Fume in air. Remarkably inert toward metals, even sodium.

SiHCl<sub>3</sub>: Synonym: silicochloroform. M.p.  $-134^{\circ}$ C, b.p.  $36.5^{\circ}$ C; d 1.35. In the presence of AlCl<sub>3</sub> disproportionates at temperatures above  $300^{\circ}$ C to SiCl<sub>4</sub> and SiH<sub>2</sub>Cl<sub>2</sub>.

SiH<sub>2</sub>Cl<sub>2</sub>: M.p. -122°C, b.p. 8.5°C; d (-122°C) 1.22. SiH<sub>3</sub>Cl: M.p. -118°C, b.p. -30.5°C; d (-113°C) 1.15. **REFERENCES:** 

L. Gatterman, Ber. dtsch. chem. Ges.  $\underline{22}$ , 190 (1889); O. Ruff and K. Albert, Ber. dtsch. chem. Ges.  $\underline{38}$ ,  $\underline{2226}$  (1905); C. Combes, Comptes Rendus Hebd. Séances Acad. Sci.,  $\underline{122}$ , 531 (1896); G. H. Wagner, U. S. Patents 2,657,114; 2,627,451; A. Stock and C. Somieski, Ber. dtsch. chem. Ges.  $\underline{52}$ , 695 (1919); D. T. Hurd, J. Amer. Chem. Soc. <u>67</u>, 1545 (1945); G. H. Wagner, U. S. Patent 2,499,009; see G. Fritz, Z. anorg. allg. Chem., <u>280</u>, 134 (1955), for quantitative determination of the Si-H bond energies.

### Tribromosilane

SiHBr<sub>3</sub>

 $Si + 3 HBr = SiHBr_3 + H_2$ 28.1 242.8 268.8 22.4 l

Standard powdered Si (100-mesh, about 97.5% Si) can be used for the preparation; the addition of some CuCl is advisable. The apparatus shown in Fig. 228 should be used. Hydrogen, dried with fused KOH in drying towers a, flows through a bubbler b which is filled with Br<sub>2</sub> and heated to 45°C on a water bath. The Br<sub>2</sub>-laden gas then flows through tube c, filled with Pt-asbestos and heated to about 200°C, where it is converted to HBr. Either FeBr<sub>2</sub> (dried at 100°C) or moist red phosphorus is placed in d. The towers e are



Fig. 228. Preparation of tribromosilane.

filled with CaBr<sub>2</sub>. The charges in d and e must be replenished frequently. Tube f contains the Si; it is heated to  $360-400^{\circ}$ C. Flask g is placed in a cooling bath at  $-30^{\circ}$ C. If lower temperatures are used, the byproduct SiBr<sub>4</sub> tends to condense and plug the tube.

Before a run, the air is displaced with  $H_2$  and the apparatus is thoroughly dried. The furnaces are then turned on and  $Br_2$  is added to the bubbler from the dropping funnel. The gas stream is so regulated that it passes through the  $Br_2$  at the rate of 3-4 bubbles per second. Under these conditions (45°C), about 60 g. vaporizes in 5 hours. Unreacted HBr and  $H_2$  leave the apparatus through the CaCl<sub>2</sub> tube t. At the end of the run, the product is distilled directly from flask g into reagent bottles *i* provided with fusible necks. When such a bottle is full, the stopcock h is closed and the bottle is sealed. When a fresh bottle is attached, it is first evacuated and then filled by opening h. These safety measures are important since SiHBr<sub>3</sub> is highly flammable and readily hydrolyzed. The distillation produces two fractions: I, boiling up to 125°C and II, boiling from 125 to 154°C.

Each of these fractions undergoes a final fractional distillation in a column. The apparatus is shown in Fig. 229. The sealed bottles i are arranged on pads of glass wool in receiving vessels. On top of the vials rest glass-encased iron hammers k; these are later lifted with magnets and used to shatter the tips of the reagent bottles (see Part I, p. 63). The apparatus is evacuated and gently heated by fanning with an open flame. The reagent bottles are now opened, flask b is cooled in a cold bath to  $-40^{\circ}$ C, and the contents of the bottles are distilled into b. The receiving vessels containing the empty reagent bottles are sealed off. The apparatus is now filled with dry N<sub>2</sub> and the distillation is started in a stream of N<sub>2</sub>.



Fig. 229. Fractionation of crude obtained in the preparation of tribromosilane.

The jacket of the column is maintained at a temperature barely under the boiling point by means of a stream of air preheated in I. The fractions boiling at 64°C, 111.8°C and 153.4°C are collected. They contain, respectively, SiH<sub>2</sub>Br<sub>2</sub>, SiHBr<sub>3</sub> and SiBr<sub>4</sub>. A total of about 57 g. of the crude is obtained from 60 g. of Br. After fractionation, about 60-70% of the crude is recovered as SiHBr<sub>3</sub>.

Purity may be determined conveniently by gasometry, according to the equation:  $SiHBr_3 + 5 NaOH = 3 NaBr + Na_2SiO_3 + 2H_2O + H_2$  or by melting-point and boiling-point determinations.

**PROPERTIES:** 

Colorless, mobile liquid. B.p.  $111.8^{\circ}$ C, m.p.  $-73.5^{\circ}$ C. Ignites easily in air. Vap. p. (0°C), 8.8 mm. Log p = -1819.5/T + 7.6079. Hydrolyzed in cold water to  $H_2Si_2O_3$  and HBr.

**REFERENCES:** 

A. Besson and L. Fournier, Comptes Rendus Hebd. Séances Acad. Sci., <u>151</u>, 1055 (1911); H. Buff and F. Woehler, Liebigs Ann. Chem. <u>104</u>, 99 (1857); C. Combes, Comptes Rendus Hebd. Séances Acad. Sci., <u>122</u>, 531 (1896); Bull. Soc. Chim. France (3), <u>7</u>, 242 (1892); L. Gatterman, Ber. dtsch. chem. Ges. 22, 193 (1889); W. C. Schumb and R. C. Young, J. Amer. Chem. Soc. 52, 1464 (1930); W. C. Schumb in H. S. Booth, Inorg. Synthesis, Vol. I (New York and London, 1939), p. 38; G. Schott and W. Herrmann, Z. anorg. allg. Chem. <u>288</u>, 4 (1956).

## Dimethyldichlorosilane

 $(CH_3)_2SiCl_2$ 

 $2 CH_{3}Cl + Si = (CH_{3})_{2}SiCl_{2}$ 101.0 28.1 129.1

Commercial grade powdered Si (170 g.) is rinsed with hydrochloric acid, placed in a Pt dish and washed with water. It is then heated with hydrofluoric acid and with sulfuric acid, again well washed with water and thoroughly dried. It is then mixed with 30 g. of copper (I) chloride. The resulting powder may be used as is. However, it is preferable to tablet it before placing it in a 40-cm.-long and 3-cm.-diameter Pyrex tube. The tube is wrapped with an asbestos layer, on top of which there is a heating coil. This assembly is then wrapped with a double layer of asbestos tape. One end is closed with a two-hole rubber stopper. A glass tube is inserted through one hole of the stopper and connected through a bubbler filled with concentrated  $H_2SO_4$  to a cylinder of methyl chloride. The other hole of the stopper is fitted with a thermometer. The other end of the tube is drawn out and slightly bent downward. It is attached to a Liebig condenser provided with a receiver. A trap, maintained at  $-80^{\circ}C$ , is attached to the receiver; and, finally, a CaCl<sub>2</sub> tube is connected to the outlet side of the trap. The entire apparatus is carefully dried prior to the run. The methyl chloride is introduced through the bubbler at a rate of 2 bubbles/second; the temperature of the tube is controlled at 300°C. Condensate immediately appears in the receiver. The run is terminated after about 70 hours; the condensates in the trap and in the receiver are combined and the mixture is fractionally distilled, using a good column. The desired product comes over at 70°C. About 30 to 40% of the crude is (CH<sub>3</sub>)<sub>2</sub>SiCl<sub>2</sub>; about the same amount of CH<sub>3</sub>SiCl<sub>3</sub> (b.p. 65.7°C) is also obtained.

Analysis:

For the determination of Si in low-boiling organic compounds, see B. Smith, Acta Chem. Scand., 11, 579 (1957).

PROPERTIES:

Colorless liquid. M.p.  $-76.1^{\circ}$ C, b.p.  $70^{\circ}$ C; d ( $25^{\circ}$ C) 1.06. Hydrolyzes in moist air or water with the formation of polymeric methylsiloxanes.

**REFERENCES:** 

F. Kraus and A. von Grosse, Die Chemie der metallorganischen Verbindungen (Berlin, 1937); E. G. Rochow, J. Amer. Chem. Soc. <u>67</u>, 963 (1945); M. J. Hunter, J. F. Hyde, E. L. Warrick and H. J. Fletcher, J. Amer. Chem. Soc. <u>68</u>, 667 (1946); W. Patnode and D. F. Wilcock, J. Amer. Chem. Soc. <u>68</u>, 358 (1946); W. Patnode, U.S. Patent 2,380,997; E. G. Rochow in L. F. Audrieth, Inorg. Syntheses, Vol. III (New York-Toronto-London, 1950), p. 56.

## Chlorosiloxanes

 $Si_4O_4Cl_8$ ,  $Si_nO_{n-1}Cl_{2n+2}$ 

Powdered silicon is heated in a 2:1 mixture (by volume) of  $Cl_2$ and  $O_2$ , using the same apparatus as for the preparation of SiCl<sub>4</sub>. Both the gases and the apparatus must be carefully dried. The  $O_2$ is introduced first, and then the  $Cl_2$  flow is started. It is best not to rely upon the observation of gas bubbles in the wash bottles for
estimating the gas flow; rather, a flow meter filled with  $H_2SO_4$  should be installed. The receiver is cooled with ice. The reaction product is separated from SiCl<sub>4</sub> by distillation up to  $80^{\circ}C$ . The Si<sub>2</sub>OCl<sub>6</sub> then comes over up to  $137^{\circ}C$ . It is then refractionated at 15 mm.

Other preparative possibilities: Partial hydrolysis of  $SiCl_4$  in ether at reduced temperatures. Heating of  $SiCl_4$  with  $SO_3$  in a sealed tube.

PROPERTIES:

 $Si_4O_4Cl_9$ ; Colorless crystals. M.p. 77°C, b.p. (15 mm.) 91°C. The other members of the series are viscous liquids which solidify as glasses. They are miscible with  $CCl_4$ ,  $CHCl_3$  and  $CS_2$  in all proportions. Form esters with absolute alcohol. The latter are quite stable and are only slightly hydrolyzed even when boiled in water.

B.p. (15 mm.)  $Si_3O_2Cl_8$ , 76°C;  $Si_4O_3Cl_{10}$ , 109–110°C;  $Si_5O_4Cl_{12}$ , 130–131°C;  $Si_6O_5Cl_{14}$ , 139–141°C;  $Si_7O_6Cl_{16}$ , 145–147°C.

**REFERENCES:** 

W. C. Schumb and D. F. Holoway, J. Amer. Chem. Soc., <u>63</u>, 2753 (1941); W. C. Schumb and A. J. Stevens, J. Amer. Chem. Soc. <u>69</u>, 726 (1947); <u>72</u>, 3178 (1950); 75, 1513 (1953); J. Goubeau and R. Warnick, Z. anorg. allg. Chem. 259, 109 (1949).

# Silicon Monoxide

SiO

A mixture of finely divided silicon (about 98.5% Si) and the purest calcined quartz powder is sublimed in a high vacuum.

$$\begin{array}{rcl} \text{Si} &+& \text{SiO}_2 &=& 2 \text{SiO} \\ 28.1 & & 60.1 & & 88.1 \end{array}$$

The intimate mixture (preferably in the form of briquettes) is placed in the closed end of a porcelain tube. The tube is attached to a high-vacuum pump. The pressure is reduced to  $10^{-3}$  to  $10^{-4}$  mm. and then the closed end of the porcelain tube is slowly heated in an electric furnace to about  $1250^{\circ}$ C. The run is complete after about 4 hours. Upon cooling, the SiO is found as a black, compact, shellaclike substance in that part of the tube which has been moderately heated. The transition zone of the tube, heated to about  $400-700^{\circ}$ C during the run, contains a brown

mass of large volume which consists of a stoichiometric mixture of Si and SiO<sub>2</sub>. The brittle SiO is easily scraped off the wall of the tube with a stainless steel spatula. Occasionally it changes spontaneously to SiO<sub>2</sub> upon exposure to the air; it is therefore desirable to empty the tube in a nitrogen atmosphere.

It is particularly important that the  $400-700^{\circ}$ C transition zone, in which the already formed SiO decomposes to Si and SiO<sub>2</sub>, be as short as possible. This is best accomplished by the use of ceramic tubes, which are poor heat conductors. On the other hand, this zone would be longer in a metal tube which is a good heat conductor. Such metal tubes were recommended in the past. However, their use may cause a very marked decrease in the yield.

Other preparative methods: Reduction of  $SiO_2$  with charcoal in a vacuum.

Calcining an intimate mixture of Si and  $SiO_2$  for 9 hours at 1300°C, followed by rapid cooling, produces cubic SiO crystals.

#### **PROPERTIES:**

Formula weight 44.06. Black to brown-black, amorphous, shellaclike substance or cubic crystals. d 2.18-2.2. Hardness is about equal to Si. Disproportionates between 400 and 700°C to  $SiO_2$  and Si. Nonconducting. M.p. > 1700°C. Produces H<sub>2</sub> with alkali hydroxides and dissolves forming silicates.

### **REFERENCES:**

K. F. Bonhoeffer, Z. physik. Chem. (A), <u>131</u>, 360 (1928); W. Blitz, Naturwissen. <u>26</u>, 188 (1938); E. Zintl, Z. anor. allg. Chem., <u>245</u>, 1 (1940); G. Grube and H. Speidel, Z. Elektrochem., <u>53</u>, 339 (1949); H. Koenig, Optik, <u>3</u>, 419 (1948); H. von Wartenberg, Z. Elektrochem., <u>53</u>, 343 (1949); M. Hoch and H. J. Johnston, J. Amer. Chem. Soc., <u>75</u>, 5224 (1953); L. Brewer and R. K. Edwards, J. Phys. Chem., <u>58</u>, 351 (1954); G. Jacobs, Comptes Rendus Hebd. Séances Acad. Sci., 236, 1369 (1953).

# Silicic Acids

There exists a series of true silicic acids with various molecular sizes. The simplest is the ortho form of monosilicic acid,  $H_4SiO_4$ . It is soluble in water and is "molybdate-active" (i.e., it immediately produces a yellow color with ammonium molybdate, a phenomenon which is used in its quantitative determination). Condensation to oligosilicic acids takes place through intermolecular dehydration when the concentration of the solution is greater than 10 mg.  $SiO_2/100$  ml.  $H_2O$ . With higher concentrations, polysilicic acids with high molecular weights and colloidal properties are formed.

# 1. AQUEOUS MOLECULAR DISPERSIONS OF SILICIC ACID

Immediately after drying, 20 g. of unpulverized silica gel, prepared by the method presented in 3 below, is shaken for 48 hours with 400 ml. of double-distilled water. Upon contact with water and with agitation, the gel crumbles into small granules. Finally it is filtered by means of an ultrafilter. The filtrate contains orthosilicic acid, the concentration of which can be determined either colorimetrically or by evaporation. The initial concentration in a solution prepared in this manner is generally between 12 and 15 mg.  $SiO_2/100$  ml. This is diluted to 10 mg.  $SiO_2/100$  ml. The filtration residue may be reused after drying at 80°C for 24 hours.

# 2. COLLOIDAL SILICIC ACID

a) A solution of 60 g. of sodium metasilicate (see p. 704) in 200 ml. of warm water is filtered. After cooling, the clear solution is poured into 100 ml. of a solution composed of equal parts of water and concentrated hydrochloric acid; the solution must be well stirred during the addition and the final solution must be acid. The solution is then dialyzed until treatment with  $AgNO_3$  produces only a slight turbidity but no permanent precipitate.

b) Silica gel (20 g.), prepared according to Paragraph 3 below, is shaken with 400 ml. of 5 N ammonia in the manner described in Paragraph 1 above; the solution is filtered and the ammonia is separated by vacuum distillation while cold. Traces of ammonia which remain in the solution stabilize the colloidal suspension.

# 3. SILICA GEL

The purest grade of silica gel is obtained by hydrolysis of tetramethoxysilane [tetramethylorthosilicate, Si(OCH<sub>3</sub>)<sub>4</sub>, the methyl ester of orthosilicic acid, b.p.  $121^{\circ}C$ ], which can be prepared by the procedure given on p. 702. If the ester contains traces of HCl, distillation over dry Ag<sub>2</sub>O is recommended. Fifty grams of the ester are added to 80 ml. of double-distilled water in a large platinum dish and heated to  $40-50^{\circ}C$  while stirring with a platinum spatula. A homogeneous mass is formed after 15 minutes. A rather voluminous gel is formed after standing for one hour in a warm place. The alcohol and most of the water are removed by drying for 48 hours at  $80^{\circ}C$ . The dried gel has the appearance of hard, pea-size and pearllike lumps.

Because of its extremely slow rate of hydrolysis, the use of tetraethylorthosilicate for this purpose is not recommended.

# 4. CRYSTALLINE DISILICIC ACID

Sulfuric acid (80%, 100 ml.) is cooled to 10°C with vigorous mechanical agitation. As soon as this temperature is attained, 5 to 10 g. of finely divided and sieved crystalline sodium disilicate is gradually added, preferably through a sieve, with vigorous stirring. Stirring with cooling is continued for at least another 3 hours. Then, 5 liters of distilled water is added with stirring to the mixture. As soon as a precipitate forms, the solution is decanted and fresh water is poured in. This is again decanted after some stirring; the water is changed in this manner 4 or 5 times. Finally, the mixture is filtered, washed until the filtrate is free of  $SO_4^{2-}$  and then rinsed with an alcohol-ether solution. The ether is removed by filtration. Water loss on ignition is about 14% [theoretical (H<sub>2</sub>Si<sub>2</sub>O<sub>5</sub>) 13.05%]. After mixing with hydrofluoric acid and evaporating, not more than 0.1% Na2SO4 should be present if the washing was thorough and treatment with acid was sufficiently long. An x-ray examination shows the substance to be crystalline.

#### **REFERENCES:**

R. Schwarz and E. Barnetzky, Angew. Chem. <u>68</u>, 573 (1956); R. Schwarz and E. Menner, Ber. dtsch. chem. Ges. <u>57</u>, 1477 (1924); R. Schwarz, Z. anorg. allg. Chem. <u>276</u>, 33 (1954); R. Schwarz and H. W. Hennicke, Z. anorg. allg. Chem. <u>283</u>, 346 (1956); private communication from R. Schwarz.

### Silicon Oxyhydride

# H<sub>2</sub>Si<sub>2</sub>O<sub>3</sub>

A mixture of equal parts of SiHCl<sub>3</sub> and benzene is poured with stirring into 10 times its volume of ice-water mixture. The SiHCl<sub>3</sub> must be carefully fractionated to free it of SiCl<sub>4</sub>. After a while, the reaction product is filtered off and purified by repeated suspension and washing with water. Complete removal of chlorine is difficult and is possible only by dialysis, as described by Schwarz and Souard. The product prepared as above still contains 0.5% Cl after three washings. It is dried overnight in a stream of air at  $125^{\circ}$ C.

#### PROPERTIES:

Fine, light, white powder; amorphous. Wiberg and Simmler describe the preparation of a crystalline substance. Flammable.

Evolves  $H_2$  when treated with alcohols. The Si-H group adds alkenes to alkyl groups and may be chlorinated. When heated above  $350^{\circ}$ C in an inert atmosphere, splits off  $H_2$  and forms  $Si_2O_3$ . At 900°C, the entire theoretical quantity of  $H_2$  is liberated in a few minutes, leaving  $Si_2O_3$ .

**REFERENCES:** 

R. Schwarz and R. Souard, Ber. dtsch. chem. Ges., <u>53</u>, 1 (1920); R. Mueller, Chem. Techn., <u>2</u>, 7, 41 (1950); G. H. Wagner and A. N. Pines, Ind. Eng. Chem., <u>44</u>, <u>321</u> (1952); E. Wiberg and W. Simmler, Z. anorg. allg. Chem., <u>283</u>, 401 (1956).

### Silicon Disulfide

SiS<sub>2</sub>

Somewhat more than the stoichiometric quantity of S is melted with 200-300 g. of aluminum pellets in a Hessian crucible. After cooling, some of the same reaction mixture is placed loosely on top of the solid mass. A strip of magnesium is inserted and ignited. After the very vigorous reaction is complete and the mixture has cooled, the contents of the crucible are finely pulverized and mixed with a little more than the theoretical amount of fine quartz sand. The mixture is placed in an unglazed porcelain quartz boat and heated in a stream of pure N<sub>2</sub> in either a porcelain or quartz tube. The reaction starts at 1100°C. Between 1200 and 1300°C, a feltlike sublimate of SiS<sub>2</sub> forms at the cooler parts of the tube. The SiS, which is deposited simultaneously, is easily separated from the SiS<sub>2</sub> since their volatilities differ.

Further purification is accomplished by resublimation in a stream of inert gas or in vacuum.

Other preparative possibilities: Passage of dry  $H_2S$  over powdered silicon at  $1200 - 1300^{\circ}C$ .

PROPERTIES:

Formula weight 92.18. White, fibrous substance. Very sensitive to moisture. M.p.  $1090^{\circ}C$ ; d 2.02. Burns slowly when heated in air.

**REFERENCES:** 

E. Tiede and M. Thiemann, Ber. dtsch. chem. Ges., <u>59</u>, 1703 (1926); E. Zintl, Z. physik. Chem. (A), <u>174</u>, 301 (1935); R. Schwarz, Z. anorg. allg. Chem., 276, 33 (1954).

### Silicon Tetraacetate

Si(CH<sub>3</sub>COO)<sub>4</sub>

A dropping funnel b is used to add 255 g. of  $SiCl_4$  to 744 g. of acetic anhydride, placed in a three-neck, one-liter flask (Fig. 230).



I.

Fig. 230. Preparation of silicon tetraacetate.

The apparatus must be completely dry and protected from atmospheric moisture. The filter c is pulled up and the tube closed by a pinch-clamp e. The reaction mixture evolves heat on agitation and crystals of Si(Ac) precipitate The mixture is allowed to at once. stand for several days. It is then cooled with Dry Ice, and filter c is lowered until it reaches the supernatant liquid, siphoned into the suction which is flask. Acetic anhydride (100 ml.) is now added to the residue in a, the crystals are dissolved by heating at 100°C and allowed to recrystallize by cooling to 0°C, and the mother liquor is removed as above. A second recrystallization proceeds in the same manner, but with 75 ml. of acetic anhydride. The solution adhering to the

crystals is removed by evaporation, first at room temperature and then at  $100^{\circ}$ C (several hours). The yield is 335 g. or about 85%, based on SiCl<sub>4</sub>.

II. Four moles of acetic acid and 1.5 moles of  $SiCl_4$  are refluxed for 48 hours in 200 ml. of absolute ether. After standing for two days in the cold, the precipitated crystals are filtered off with exclusion of moisture. An additional yield is obtained by concentration of the mother liquor. The yield is 70 g., or about 30%, based on acetic acid.

Other possible preparative methods: Reaction of  $SiCl_4$  with thallium (I) acetate. The procedure is described in detail on p. 726 (section on Germanium).

PROPERTIES:

Very hygroscopic. Effervescent reaction with water, producing  $SiO_2$ . Decomposes on heating at 160-170°C. Forms ethyl acetate

and SiO<sub>2</sub> with alcohol. M.p.  $110^{\circ}$ C, b.p. (5-6 mm.) 148°C. Soluble in acetone and benzene.

**REFERENCES:** 

C. Friedel and A. Ladenburg, Liebigs Ann. Chem., <u>145</u>, 174 (1868); J. H. Balthis in J. C. Bailar, Inorg. Synthesis, Vol. IV (New York-London-Toronto, 1953), p. 45, J. Goubeau and R. Mundiel, Z. anorg. allg. Chem., <u>272</u>, 313 (1953); H. Schmidt, C. Blohm and G. Jander, Angew. Chem. 59, 235 (1947).

# Silicon Cyanate and Silicon Isocyanate

### Si(OCN)<sub>4</sub>, Si(NCO)<sub>4</sub>

 $SiCl_4 + 4 AgNCO = Si(NCO)_4 + 4 AgCl_{169.9} 599.6 196.2 573.2$ 

A little less than the stoichiometric quantity of SiCl<sub>4</sub>, dissolved in dry benzene, is added with stirring to a suspension of silver isocyanate, also in dry benzene. The solution is filtered and distilled at 25 mm. The lower-boiling fraction contains the Si(NCO)<sub>4</sub>; the higher boiling fraction contains the Si(OCN)<sub>4</sub>. The yield of the latter, however, is only about 2.5%.

PROPERTIES:

Colorless crystalline substances. Si(NCO) 4: B.p. 185.6°C, m.p. 26.0°C; d (20°C) 1.413. Si(OCN) 4: B.p. 247.2°C, m.p. 34.5°C; d (20°C) 1.414.

**REFERENCE:** 

G. S. Forbes, J. Amer. Chem. Soc., 62, 761 (1940).

### Tetraethoxysilane, Tetramethoxysilane

#### $Si(OC_2H_5)_4$ , $Si(OCH_3)_4$

Dehydrated alcohol is distilled from CaO and metallic Ca (or  $CaC_2$ ) into a one-liter, three-neck flask which has been carefully

702

predried. A mercury seal stirrer is mounted in the center tube of the flask, and the third tube is fitted with a reflux condenser equipped with a CaCl<sub>2</sub> tube and an exit tube.

When the flask is about half full (about 400 g. of alcohol), the adapter through which the alcohol was introduced is replaced by a dropping funnel. The end of the funnel tube should be several cm. below the surface of the alcohol. The flask is placed in a cold-water bath. The stirrer is now started and the SiCl, slowly added from the funnel. Over the course of several hours about 10%less SiCl<sub>4</sub> than that required by the above equation is allowed to flow in, that is, about 330 g. of the ethoxide or 230 g. of the methoxide. Toward the end of the reaction, the liquid foams considerably. Stirring is continued for an additional half hour, and the flask is then slowly heated. Most of the dissolved HCl is volatilized during this procedure and the solution is finally brought to a boil. When no further HCl evolves, the solution is allowed to cool somewhat; the reflux condenser is replaced by a distilling condenser, the excess alcohol is distilled off and the flask is heated to about 100°C. After cooling, some sodium alcoholate (prepared by dissolving some Na in alcohol dried over Ca) is added; the solution is shaken and is then allowed to settle. After several hours, the clear liquid is decanted into a dry distilling flask and fractionated. The yield is 90%.

The yield depends solely on the degree to which moisture is excluded and on the water content of the alcohol. If, for example, 96% alcohol is used instead of absolute alcohol, the principal product will be esters of disilicic acid and polymeric metasilicic acid. Esters of other alcohols can be obtained in the same manner; in the preparation of the methyl ester, however, special care must be taken to exclude moisture, and perfectly dry methanol must be used.

SYNONYM::

Alkyl ester of silicic acid.

PROPERTIES:

Colorless liquids. B.p.  $165^{\circ}$ C or  $121^{\circ}$ C, respectively. Immiscible with water; slowly hydrolyze.

#### **REFERENCES:**

P. A. Thiessen and O. Koerner, Z. anorg. allg. Chem. <u>189</u>, 168 (1930); P. W. Schenk, unpublished research.

# Silicates

SODIUM METASILICATE AND SODIUM DISILICATE, Na2SiO3 AND Na2Si2O5

Very pure quartz sand is intimately mixed with Na<sub>2</sub>CO<sub>3</sub> (or NaHCO<sub>3</sub>) in the appropriate proportions and the mixture is melted in a platinum crucible at 1150°C. Since the metasilicate is especially difficult to obtain in crystalline form, the glassy solid melts must be maintained for considerable time at 700°C (about 100 hours is required for Na<sub>2</sub>SiO<sub>3</sub> and 50 hours for Na<sub>2</sub>Si<sub>2</sub>O<sub>5</sub>).

#### PROPERTIES:

M.p.: Na<sub>2</sub>SiO<sub>3</sub>, 1089°C; Na<sub>2</sub>Si<sub>2</sub>O<sub>5</sub>, 874°C. Soluble in water, but the solution is not clear because of SiO<sub>2</sub> precipitation.

**REFERENCES:** 

R. Schwarz, Z. anorg. allg. Chem., <u>126</u>, 62 (1923); R. Schwarz and E. Menner, Ber. dtsch. chem. Ges., <u>57</u>, 1477 (1924).

Na<sub>2</sub>SiO<sub>3</sub> · 9 H<sub>2</sub>O

$$\begin{array}{rcrcr} Na_{2}SiO_{3} + 9 H_{2}O = Na_{2}SiO_{3} \cdot 9 H_{2}O \\ 122.1 & 162.0 & 284.1 \end{array}$$

One part by volume of waterglass (analysis: 35.8 g. SiO<sub>2</sub> and 11.3 g. Na<sub>2</sub>O per 100 ml.) is mixed with one part by volume of H<sub>2</sub>O and two parts by volume of sodium hydroxide solution (sp. g. 1.27) and allowed to stand. (If there is substantial deviation from the given analysis of the waterglass, the proportion of caustic soda must be adjusted.) The solution is seeded, using seed crystals obtained from precipitation of a part of the solution with alcohol. When the white mass solidifies after several hours, the mother liquor is removed by compression, by suction, or, best, by centrifugation. The product should be recrystallized once or twice from warm 2-3% sodium hydroxide. (Retain some seed crystals.) Finally the precipitate is dried on filter paper.

#### PROPERTIES:

Colorless, very slightly efflorescent salt in the form of small platelike crystals. M.p. 48°C in its water of crystallization.

REFERENCES:

K. Vesterberg, Z. anorg. allg. Chem., <u>88</u>, 341 (1914); H. Lange and M. von Stackelberg, Z. anorg. allg. Chem. 256, 271 (1948).

LITHIUM METASILICATE, Li2SiO3

 $Li_2CO_3 + SiO_2 = Li_2SiO_3 + CO_2$ 73.9 60.1 89.9 44.0

Lithium metasilicate can be prepared in the same way as sodium metasilicate by melting stoichiometric quantities of very pure  $\text{Li}_2\text{CO}_3$  and quartz sand in a platinum crucible.

PROPERTIES:

Colorless, crystalline substance. M.p. 1201°C. The precise melting point and its excellent crystalline characteristics make it suitable for use as a calibration standard for thermocouples.

**REFERENCES:** 

R. Schwarz and H. Sturm, Ber. dtsch. chem. Ges., <u>67</u>, 1737 (1914); F. C. Kracek, J. Phys. Chem., 34, 2641 (1930).

LEAD METASILICATE, PbSiO<sub>3</sub>

 $\begin{array}{rrr} Pb(NO_3)_2 &+ Na_2SiO_3 &= PbSiO_3 &+ 2 NaNO_3 \\ 331.2 & 122.1 & 283.3 & 170.0 \end{array}$ 

Sodium metasilicate, prepared as above from  $SiO_2$  and  $Na_2CO_3$ , is dissolved in water to obtain a 0.2% solution; a 0.5% lead nitrate solution is added with stirring to this solution. After settling, the precipitate is easily filtered off. It is dried on a water bath. A dark-colored preparation can be avoided only by carrying out the procedure in an atmosphere completely free of  $H_2S$ .

PROPERTIES:

White powder; m.p. 780°C.

**REFERENCES:** 

R. Schwarz, Z. anorg. allg. Chem., 126, 76 (1923).

Lemon-yellow, light-sensitive silver metasilicate  $Ag_2SiO_3$  can be obtained in the same manner [R. Schwarz, Z. anorg. allg. Chem., 126, 76 (1923); Z. Elektrochem. 32, 415 (1926)]. BARIUM METASILICATE,  $BaSiO_3$ , AND BARIUM DISILICATE,  $BaSi_2O_5$ 

A solution of about 1% sodium silicate is prepared and, after filtering, precipitated by adding 2%  $BaCl_2$  solution with stirring; the  $BaCl_2$  should be in 10% excess. The precipitate is dried by suction and then on a water bath. This yields the disilicate. The metasilicate is formed if NaOH is added to the silicate solution before precipitation so that there are fifteen moles of base to one mole of SiO<sub>2</sub>. This is dried in the same manner as above.

REFERENCE:

R. Schwarz and H. Richter, Ber. dtsch. chem. Ges., 60, 2269 (1927).

# Germanium

The starting material for the preparation of germanium and its compounds in the laboratory is germanite, a copper thiogermanate found at Tsumeb in Southwest Africa. A series of procedures has been developed for laboratory preparation, of which both acid and alkaline decompositions have proved to be suitable. Recovery of germanium by sublimation of GeS in a stream of  $NH_3$  requires somewhat more equipment. Alkaline decomposition is especially recommended when it is also desired to recover the approximately 0.9% of gallium contained in the germanite. All the recovery methods proceed through initial formation of the dioxide,  $GeO_2$ .

## Germanium (IV) Oxide

#### GeO<sub>2</sub>

#### I. ACID DECOMPOSITION

One liter of water and half a liter of concentrated  $H_2SO_4$  are poured over 500 g. of pulverized ore in a 5-liter Erlenmeyer flask, which is placed in a large evaporating dish under an efficient hood or, better still, in the open air. The mixture is shaken and then 1.2 liters of concentrated nitric acid (d 1.4) is added all at once. When the very vigorous reaction ends (foaming over occurs only with too highly concentrated acid), the mixture is heated for several hours on the water bath until the  $NO_2$  evolution ceases almost completely. The hot solution is filtered rapidly through a suction filter and the filtrate is poured into a flat dish. Copper sulfate crystallizes out overnight. After decanting from the CuSO<sub>4</sub>, the filtrate is reused for processing another 500 g. of ore, for which purpose 200 ml. of concentrated  $H_2SO_4$  and 1 liter of nitric acid are added to it. The residue on the filter is crude GeO<sub>2</sub> contaminated with SiO<sub>2</sub>, and it is purified in the following way.

The crude  $GeO_2$  from 2-4 decomposition runs is placed in a 2-liter Pyrex flask (a in Figure 231) provided with a ground-glass joint, and hydrochloric acid is added. Each decomposition of 500 g.



Fig. 231. Distillation of germanium tetrachloride.

of ore yields about 30 g. of  $\text{GeO}_2$ . One run consumes 210 ml. of concentrated hydrochloric acid (d 1.19), so that 840 ml. of the 28% technical grade hydrochloric acid is used with the  $\text{GeO}_2$  from four decomposition runs. The receivers are then cooled and the flask is very slowly heated while passing through a slow stream of Cl<sub>2</sub>. The receiver filled with water should not be allowed to become warm. If this occurs, either too much HCl was added or the flask was heated too quickly, that is, before the equilibrium

$$GeO_2 + 4 HCl \rightleftharpoons GeCl_4 + 2 H_2O$$

had been established. The GeCl<sub>4</sub>, which is yellow because of the  $Cl_2$ , distills over at 83°C. A second layer, which consists of 20% hydrochloric acid, collects over it later. Toward the end of the

distillation, GeO<sub>2</sub> sometimes settles out in the receiver as a result of hydrolysis. This occurs if too little or too dilute hydrochloric acid has been used. In this case, concentrated hydrochloric acid from a separatory funnel is added to the liquid in the flask. If the receiver containing water should become warm at any time, it should be cooled with ice and, if necessary, the water should be changed. Distillation is carried out until the liquid in the first receiver becomes completely clear and no further GeCl<sub>4</sub> collects there. The clear distillate, which settles out in a fresh receiver, is tested by introducing H<sub>2</sub>S, after reducing the dissolved Cl<sub>2</sub> with H<sub>2</sub>SO<sub>3</sub>. No white GeS<sub>2</sub> should precipitate. A quicker test for Ge (without removing the Cl<sub>2</sub>) is also possible if the solution is somewhat diluted and a 2% tannic solution is added [G. Brauer and H. Renner, Z. analyt. Chem. 133, 401 (1951)].

# II. ALKALINE DECOMPOSITION

A solution of 500-600 g. of NaOH in an equal quantity of water is prepared and placed in a steel crucible. Then 1 kg. of very finely divided germanite (comminuted in a ball mill) is slowly added to the hydroxide solution. Vigorous agitation with an iron spatula is essential. Sometimes it is also necessary to cool the reaction vessel intermittently with cold water. After completion of the addition the mixture is slowly heated while the agitation is maintained. The heating is continued until the mixture becomes either highly concentrated (thick) or almost dry. At this point, it is scraped out of the crucible and transferred to a flask containing one liter of water. The resulting solution is filtered immediately thereafter (Pyrex glass frit or asbestos paper). The residue on the filter is thoroughly washed until the wash water becomes pale yellow. These procedures yield a total of four liters of filtrate. The residue on the filter, which oxidizes very readily and thus becomes very hot in air, is discarded. It should be noted that the steps subsequent to the solution of the alkaline reaction product in water must follow each other in quick succession to avoid partial oxidation of the product; this leads to formation of colloids, which are difficult to filter. The yellow filtrate is almost neutralized with sulfuric acid and then acidified with nitric acid. After completion of the rather vigorous reaction the brownish foam, which consists chiefly of arsenic sulfide and S, is filtered off. It is processed by evaporating with concentrated  $H_2SO_4$  and  $HNO_3$  in two separate procedures. The filtrate is now neutralized with ammonia using benzylanilineazobenzene sulfonic acid (benzyl orange) as an indicator. The precipitated oxides and hydroxides of Ga and Ge are filtered off. The filtrate is treated with 10% of its volume of concentrated ammonia, and some MgSO<sub>4</sub> is added, whereby the remainder of the germanium is precipitated. A short evaporation

of the gallium-germanium precipitate in contact with  $H_2SO_4$ follows, and after dilution, the mixture is filtered. What remains is GeO<sub>2</sub>, which is purified by distillation together with the residue containing Mg. Since the material is already almost completely free of As, the purification described further below can be carried out immediately. However, the brownish foam, which is GeO<sub>2</sub> contaminated with As, must be distilled twice. The GeCl<sub>4</sub> obtained after the first distillation is hydrolyzed, and the GeO<sub>2</sub> obtained is purified as described further below (III).

To recover the valuable Ga, the following procedure is used. The sulfuric acid solution obtained after filtering off the GeO<sub>2</sub> is made alkaline and treated with some Na<sub>2</sub>S. It is then acidified, whereby As, W and Mo are precipitated. The precipitate entrains some Ga with it. It is therefore boiled with HCl. The solution is then almost neutralized with ammonia and precipated with ammonium carbonate. The Ga(OH)<sub>3</sub> is dissolved in hydrochloric acid, treated with bromine water, and boiled. The Na<sub>2</sub>S precipitate the Pb. The solution is now repeated in order to precipitate the Pb. The solution is then acidified in order to separate out the remaining Mo, and finally boiled and precipitated with ammonium carbonate. For additional processing of the crude Ga(OH)<sub>3</sub> thus obtained, see the section on Gallium.

Using either of the decomposition methods, more than 90% of the Ge contained in the ore is recovered. About 60 g. of  $\text{GeO}_2$  is usually obtained from 1 kg. of ore, while the alkaline decomposition yields, in addition, about 8-9 g. of Ga. Both methods yield an impure material. In the acid decomposition method, crude  $\text{GeCl}_4$ , already freed to a considerable extent of As, is obtained. From this,  $\text{GeO}_2$  can be obtained by hydrolysis. The alkaline decomposition method yields an arsenic-free but otherwise still rather impure  $\text{GeO}_2$ .

The following method of purification is used to prepare very pure  $GeO_2$ , in which the impurities just barely show on spectroscopic examination (total impurity level well below 0.01%).

# III. PURIFICATION OF GERMANIUM OXIDE

The combined distillates of the first distillation or the crude  $GeO_2$  from the alkaline decomposition are added to the flask of the carefully cleaned distillation apparatus described under I (for preparation of especially high purity material, a second apparatus, which is used only for this purpose, is kept on hand). The flask is half filled with pure, 6N hydrochloric acid. If the GeO<sub>2</sub> to be processed comes from alkaline decomposition, then 700 ml. of pure, concentrated, 38% hydrochloric acid (d 1.19) is added for each 100 g. of crude GeO<sub>2</sub>, and the flask may then be filled to the half-way mark with 6N hydrochloric acid. Distillation proceeds

slowly in a stream of Cl<sub>2</sub>, as described under I. Pure distilled water is added to the third receiver. The first two receivers are cooled as described under I. During the distillation, distilled water is frozen to an ice slurry in a carefully cleaned 3-liter glass flask by cooling and shaking (seed with a crystal of ice). One liter of distilled water is used for each 1.5 kg. of ore processed. The ice should crystallize out, as much as possible, during the shaking after inoculation, so that a thick crust of ice does not form on the walls of the flask, since that might lead to cracking of the flask. When a rather thick slush is obtained, the GeCl, collected in the receiver is poured into the ice slurry with vigorous shaking. The shaking is continued for 5-10 minutes, and the flask is then allowed to stand for several hours with occasional shaking. The GeO<sub>2</sub> settles out overnight. The strongly acid supernatant liquid is decanted and may be reused for diluting the concentrated hydrochloric acid in further runs. (If there is no need for this, the Ge contained in the liquid may be precipitated with H<sub>2</sub>S in the form of GeS<sub>2</sub>, following the removal of dissolved Cl<sub>2</sub> with H<sub>2</sub>SO<sub>2</sub>. However, the acid content of the solution must then be brought to 6N.) After decanting the acid supernatant liquid, the GeO<sub>2</sub> is covered with distilled water and allowed to stand for about 2-3 hours, with occasional shaking during the first hour. This treatment serves to remove the tenaciously clinging Cl ions. If this is not done, losses are incurred on later drying. The solid is dried on a fritted-glass suction filter and washed thoroughly with cold water. The GeO<sub>a</sub> is dried in an oven at 200°C while still on the filter. The filtrates and the washing water may be reused for the hydrolysis operation in additional runs. If they are not needed for this purpose, they may be worked up together with the first hydrolysis filtrate to recover the GeS<sub>2</sub>.

Further purification is possible before the hydrolysis by extracting the  $GeCl_4$  with 6N HCl according to the method of Allison and Müller or by fractional distillation in a quartz column according to the method of Green and Kafalas.

#### **PROPERTIES:**

Formula weight 104.6. Exists in one amorphous and two crystalline forms. The amorphous form always arises on cooling a melt of either of the crystalline forms and resembles a clear, strongly light-refracting glass (d 3.637). The GeO<sub>2</sub> formed by hydrolysis of GeCl<sub>4</sub> or by decomposition of germanates is of the hexagonal (quartz type) form (d 4.228). M.p. 1115°C. Solubility: 0.435 g. GeO<sub>2</sub>/100 g. H<sub>2</sub>O. The tetragonal (rutile) modification, which is almost completely insoluble in water, is obtained either by heating the GeO<sub>2</sub> at 350°C for several hours with water under pressure or by adding some ammonium fluoride to an aqueous solution of  $\text{GeO}_2$  and slowly evaporating. It is then heated for several hours at 380°C (d 6.239). M.p. 1086°C. The  $\text{GeO}_2$  preparations obtained by the hydrolysis of  $\text{GeCl}_4$  contain Cl ions even after thorough washing with water. The last traces of Cl disappear only on ignition. Pure uncalcined preparations can be obtained only by hydrolysis of a tetraalkoxygermanium. Readily soluble in strong bases and in strong hydrochloric acid.

REFERENCES:

R. Schwarz, P. W. Schenk and H. Giese, Ber. dtsch. chem. Ges. 64, 1828 (1931); R. Schwarz and E. Huf, Z. anorg. allg. Chem. 203, 188 (1931); E. R. Allison and J. H. Müller, J. Amer. Chem. Soc. 54, 2833 (1932); W. C. Johnson, S. Foster and C. A. Kraus, Ibid. 57, 1828 (1935); F. Sebba and W. Pugh, Ibid. <u>59</u>, 1371 (1937); R. Schwarz and E. Haschke, Z. anorg. allg. Chem. <u>252</u>, 170 (1943); H. J. Cluly and R. C. Chiruside, J. Chem. Soc. (London) 1952, 2275; W. Fischer and W. Harre, Angew. Chem. <u>66</u>, 165 (1954); G. H. Morrison, E. G. Dorfman and J. F. Cosgrove, J. Amer. Chem. Soc. <u>76</u>, 4236 (1954); M. Green and J. A. Kafalas, J. Chem. Soc. (London) <u>1955</u>, 1604; personal communications from G. Brauer and P. W. Schenk. [For determination of As in Ge, see S. T. Payne, Analyst <u>77</u>, 278 (1952). ]

# Germanium (II) Oxide

GeO

 $\begin{array}{rcl} {\rm GeO_2} \ + \ {\rm H_3PO_2} \ = \ {\rm GeO} \ + \ {\rm H_3PO_3} \\ {\rm 104.6} & {\rm 66.0} & {\rm 88.6} & {\rm 82.0} \end{array}$ 

A solution of 6 g. of  $GeO_2$  in about 30 ml. of strong sodium hydroxide is prepared, and enough 6N HCl is added to just redissolve the precipitate initially formed. Then 600 ml. of concentrated hydrochloric acid and 45 ml. of 50%  $H_3PO_2$  are added with cooling. The solution is heated at 100°C for 5-6 hours under a  $CO_2$  blanket, cooled and then heated with excess aqueous ammonia to precipitate the GeO. The precipitate is either filtered or centrifuged under  $N_2$ , washed and dried in a vacuum.

Other preparative methods:

$$Ge + GeO_2 = 2 GeO$$

The starting materials must be finely ground, intimately mixed and pressed into pellets [A. W. Laubengayer in L. M. Dennis, Z. anorg. allg. Chem. <u>174</u>, 107 (1928)].

A method of preparation given by E. Gastinger goes back to the same basic reaction [Z. anorg. allg. Chem. 285, 103 (1956)], in

which Ge powder is oxidized with  $CO_2$  or air under reduced pressure. The suitable apparatus is described under gallium (I) sulfide (see Fig. 250, p. 853). Boat s is filled with Ge powder; the GeO sublimes at 800 to 900°C in a stream of  $CO_2$  at atmospheric pressure, or in a stream of air at 4 to 34 mm. It precipitates on the cold finger.

PROPERTIES:

Yellow when freshly precipitated, becomes brown on boiling with water. Easily oxidized. Usually contains a few percent of  $GeO_2$ .

**REFERENCES:** 

L. M. Dennis and R. E. Hulse, J. Amer. Chem. Soc. <u>52</u>, 3553 (1930); H. M. Powell and F. M. Brewer, J. Chem. Soc. (London) <u>1938</u>, 197; W. L. Jolly, Thesis, 1952, University of California, UCRL-1638.

### Metallic Germanium

Ge

 $\begin{array}{rcl} \text{GeO}_2 &+ & 2 \text{ H}_2 &= & \text{Ge} &+ & 2 \text{ H}_2\text{O} \\ 104.6 & & 44.8 \ l & & 72.6 & & 36.0 \end{array}$ 

An unglazed porcelain boat, filled with uncalcined  $GeO_2$ , is placed in a porcelain or quartz tube heated in an electric oven at about 600°C, and a vigorous stream of H<sub>2</sub> is passed over it. The outlet end of the tube is slightly bent downward so that condensed water does not back up into the hot tube. With small quantities (under 40 g.), reduction is complete after 3-5 hours. Larger amounts necessitate longer reduction times. The reduction temperature must be watched closely, particularly at the beginning, so it does not exceed the limit. When this happens, sintering may take place and under some conditions GeO may even vaporize, in which case it deposits on the cooler portion of the tube in tuftlike crystals. Only eventually is it reduced to Ge. The Ge is allowed to cool in a stream of H<sub>2</sub>. The powdery, gray-black Ge can be melted down in a stream of H<sub>2</sub>, using an unglazed porcelain crucible with perforated cover (Rose crucible). If a compressor is used, some O<sub>2</sub> must be mixed with the compressed air in order to attain the melting temperature of the Ge. The crucible may occasionally crack because of the expansion of the Ge on hardening. It is therefore advisable to use tubular crucibles (Tammann

712

crucibles), which are tipped into an almost horizontal position shortly before the hardening of the Ge. The otherwise rather frequent cracking of the crucible is thus definitely avoided.

The melting of the Ge can also be carried out under a layer of table salt.

Other preparative methods: Reduction with powdered charcoal or KCN.

The purest metal is obtained by reduction with  $H_2$  and melting down in a stream of  $H_2$ . Yield: 90%. If the Ge adhering to the crucible wall and to the reaction tube is recovered as GeCl<sub>4</sub> by chlorination (see GeCl<sub>4</sub>), the yield becomes almost quantitative. The material obtained via this procedure is very pure: the only impurities present are those which are occasionally introduced by the crucible material. With pure raw material, spectroscopic examination shows them to be less than 0.01%.

PROPERTIES:

Brittle, shiny metal. Crumbles on a light blow with a hammer. Somewhat deliquescent on long exposure to moist air. In compact form, acid insoluble; soluble as a powder in ammonia plus  $H_2O_2$ . Burns when brought to red glow, releasing thick brown fumes of GeO. Crystals: A4 structure type (diamond). M.p. 959°C. Used in the communications industry as a detector (tuning) material.

**REFERENCES:** 

R. Schwarz and G. Elstner, Z. anorg. allg. Chem. 217, 289 (1934); private communications from P. W. Schenk and G. Brauer.

# Germanium Hydrides

GeH<sub>4</sub> (Ge<sub>2</sub>H<sub>6</sub>, Ge<sub>3</sub>H<sub>8</sub>)

Just as silicon hydrides germanium hydrides, can be prepared either via acid decomposition of magnesium germanide or via the reduction of GeCl<sub>4</sub> with LiAlH<sub>4</sub>; however, only modest yields (20-30%) can be obtained with the latter reaction because of the formation of metallic germanium. The acid decomposition of Mg<sub>2</sub>Ge with aqueous hydrochloric acid gives a smaller yield than the corresponding reaction with Si, but, just as in the case of silicon, the higher hydrides such as Ge<sub>2</sub>H<sub>6</sub> and Ge<sub>3</sub>H<sub>8</sub> can be isolated. The decomposition of Mg<sub>2</sub>Ge with NH<sub>4</sub>Br in liquid NH<sub>3</sub> gives better yields. However, the yields are again smaller than with Si. As in the latter case, primarily GeH<sub>4</sub> is formed.

Finely powdered magnesium germanide is placed in reaction vessel  $\alpha$  (Fig. 232). A 50% excess of NH<sub>4</sub>Br is placed in flask b. Then enough pure NH<sub>3</sub> is condensed in tube c so that  $\alpha$  is twothirds filled. A slow stream of NH<sub>3</sub> is allowed to continue to flow in order to provide agitation and NH<sub>4</sub>Br is slowly added from bby turning the little flask upward. The reaction begins immediately,



Fig. 232. Preparation of germanium hydrides.

with liberation of gas. The gases pass through the mercury trap f at a gage pressure of about 250 mm. and are collected over water in e. The water can be changed using flasks i and j. Toward the end, as the reaction slows down, the remainder of the NH<sub>4</sub>Br is added. The reaction takes a few hours. The NH<sub>3</sub> is allowed to vaporize and is absorbed in collecting vessel e. The reaction vessel is then heated, whereby some additional gas is evolved. From the collecting vessel the gas is pumped (by the Toepler pump l) through a P<sub>2</sub>O<sub>5</sub> drying tube k, which retains the remainder of the NH<sub>3</sub> and H<sub>2</sub>O, into the collecting tube m, where it is stored. Yield of crude germanium hydride, based on the initial Ge, is 60-70%.

The purification of the gas and the preparation of pure  $\text{GeH}_4$ can be carried out either through fractional distillation in a Stock vacuum apparatus (whereby the small amount of  $\text{Ge}_2\text{H}_6$  can be isolated), or by preparing NaGeH<sub>3</sub> or Na<sub>2</sub>GeH<sub>2</sub> by the reaction of crude  $\text{GeH}_4$  with a solution of Na in Hiquid NH<sub>3</sub>. Pure  $\text{GeH}_4$  can then be obtained by the reaction NH<sub>4</sub>Br with NaGeH<sub>3</sub> or Na<sub>2</sub>GeH<sub>2</sub>. The higher germanium hydrides  $\text{Ge}_2\text{H}_6$  and  $\text{Ge}_3\text{H}_6$  are best prepared via the decomposition of Mg<sub>2</sub>Ge with aqueous hydrochloric acid. The total yield is smaller, but the yield of  $Ge_2H_6$  and  $Ge_3H_3$  is greater.

PROPERTIES:

Colorless gases. GeH<sub>4</sub>; m.p.  $-165^{\circ}$ C, b.p.  $-88^{\circ}$ C, d ( $-142^{\circ}$ C) 1.523. Ge<sub>2</sub>H<sub>8</sub>; m.p.  $-109^{\circ}$ C, b.p. 29°C, d ( $-109^{\circ}$ C) 1.98. Ge<sub>3</sub>H<sub>8</sub>: m.p.  $-106^{\circ}$ C, b.p. 110.5°C, d 2.2. Decompose in air, often bursting into flames. Decompose to the elements above 350°C.

**REFERENCES:** 

L. M. Dennis, R. B. Corey and R. W. Moore, J. Amer. Chem. Soc. <u>46</u>, 657 (1924); C. A. Kraus and E. S. Carney, J. Amer. Chem. Soc. <u>56</u>, 765 (1934); A. E. Finholt, A. C. Bond, K. E. Wilzbach and H. J. Schlesinger, J. Amer. Chem. Soc. <u>69</u>, 2692 (1947); K. Clusius and F. Faber, Angew. Chem. <u>55</u>, 97 (1942); W. C. Johnson and S. Isenberg, J. Amer. Chem. Soc. <u>57</u>, 1349 (1935).

# Germanium (IV) Chloride

GeCl₄

 $Ge + 2 Cl_2 = GeCl_4$ 72.6 141.8 214.4

A boat with powdered Ge is placed in the tube used for the reduction of GeO<sub>2</sub> and a fast stream of dry  $Cl_2$  is passed over it while the apparatus is heated. The final temperature is 500-600°C. The reaction begins, however, at a much lower temperature, and a Pyrex reaction tube is usually sufficient. The tube can also be filled with broken boats and crucibles, to which Ge still adheres, in order to recover this germanium, which is difficult to reclaim in any other way. The tube, tilted somewhat toward the exhaust end, is connected with a gas trap by means of an adapter. The trap is cooled with ice-salt or, still better, with a Dry Ice-acetone mixture. The distillate, colored yellow by the  $Cl_2$ , is largely freed of the  $Cl_2$  by fractionation and is then redistilled over Cu powder, Hg or Hg<sub>2</sub>Cl<sub>2</sub>.

 $\begin{array}{rcl} \text{GeO}_2 \ + \ 4 \ \text{HCl} \ = \ \text{GeCl}_4 \ + \ 2 \ \text{H}_2\text{O} \\ 104.6 & 145.9 & 214.4 & 36.0 \end{array}$ 

The apparatus and the procedure are the same as used for the purification of the crude germanium oxide. However, no  $Cl_2$  is added since the starting  $GeO_2$  is arsenic-free. It is, of course,

also possible to use the unhydrolyzed  $\text{GeCl}_4$  obtained from the purifying distillation. It may be freed almost completely of dissolved  $\text{Cl}_2$  by simple distillation, with the remaining chlorine removed by shaking with copper powder. The distilled  $\text{GeCl}_4$  is separated from the hydrochloric acid in a separatory funnel, dried with ignited  $\text{Na}_2\text{SO}_4$ , and decanted from the sediment on the bottom into a distilling flask. After the first fraction has been distilled, some Cu powder or Cu turnings are added: the mixture is shaken for some time and distilled into a well-dried ampoule. The remaining Ge can be recovered as  $\text{GeO}_2$  from the first distillate and from the  $\text{Na}_2\text{SO}_4$ .

III. The GeO<sub>2</sub> (10.4 g.) is heated for eight hours at 170 to  $180^{\circ}$ C, together with ten times its weight of HCl (d 1.19), using a pressure flask placed in an oil bath. The GeO<sub>2</sub> dissolves. After cooling, the GeCl<sub>4</sub>, which separates as the heavier layer below the hydrochloric acid, is drained off in a separatory funnel and distilled.

The GeO<sub>2</sub>, which adheres tenaciously to the vessels and other apparatus, can be easily dissolved out with strong sodium hydroxide. After acidification, it can be recovered as GeS<sub>2</sub> by introducing  $H_2S$ .

PROPERTIES:

Water clear, mobile liquid. Fumes in air and is easily hydrolyzed by  $H_2O$ . Peculiar odor, easily distinguished from the simultaneously perceptible odor of HCl. M.p. -49.5°C, b.p. 83.1°C; d (19.5°C) 1.886.

**REFERENCES:** 

I and II. See references under GeO<sub>2</sub>.

III. H. Bauer and K. Burschkies, Ber. dtsch. chem. Ges. <u>66</u>, 277 (1933); A. W. Laubengayer and D. L. Tabern, J. Phys. Chem. <u>30</u>, 1947 (1926).

# Germanium Dichloride

GeCl<sub>2</sub>

 $GeCl_4 + Ge = 2 GeCl_2$ 214.4 72.6 287.0

Germanium powder, which has been thoroughly reduced in a stream of  $H_2$  at the lowest possible temperature and cooled in a stream of  $H_2$ , is placed in the small tube g of the apparatus shown in Fig. 233. The small flask a is filled with pure GeCl<sub>4</sub> (dried overnight over CaCl<sub>2</sub> and ignited Na<sub>2</sub>CO<sub>3</sub> and distilled into a).



Fig. 233. Preparation of germanium dichloride.

Flask a is then cooled with liquid nitrogen, the apparatus is evacuated, and g is heated with a small electric furnace. When greaches 300°C, c and d are opened and h is closed. Now b is cooled instead of a. The temperature of g is then elevated to 340°C. The reaction begins at 350°C. After all material distills into b, both c and d are closed and the material is redistilled into a via e. This is repeated until sufficient GeCl<sub>2</sub> is obtained in f.

A modified arrangement for the convenient preparation of  $GeCl_2$  from  $Ge + GeCl_4$  has been described by R. Schwarz and E. Baronetzki.

 $\begin{array}{rcl} \text{GeHCl}_{3} = & \text{GeCl}_{2} + & \text{HCl} \\ 180.0 & & 143.5 & & 36.5 \end{array}$ 

Even at  $-30^{\circ}$ C there is an equilibrium between considerable quantities of HCl and GeCl<sub>2</sub> and GeHCl<sub>3</sub>. Therefore, GeCl<sub>2</sub> can be obtained by pumping out the GeHCl<sub>3</sub> at  $-30^{\circ}$ C. In addition to HCl, some GeHCl<sub>3</sub> also distills into a second, liquid -nitrogen-cooled trap. Pure GeCl<sub>2</sub> remains in the first trap.

#### PROPERTIES:

Colorless, highly reactive. Begins to decompose slightly above room temperature, acquiring a color ranging from yellow to red, giving off  $\text{GeCl}_4$  and yielding chlorine-poor germanium chlorides. At an elevated temperature these finally convert to Ge. Soluble in benzene and ether.

#### **REFERENCES:**

- I. L. M. Dennis and H. L. Hunter, J. Amer. Chem. Soc. <u>51</u>, 1151 (1929); R. Schwarz and E. Baronetzki, Z. anorg. allg. Chem. <u>275</u>, 1 (1954).
- II. C. W. Moulton and J. G. Miller, J. Amer. Chem. Soc. <u>78</u>, 2702 (1956).

# Germanium (IV) Bromide

GeBr<sub>4</sub>

 $\begin{array}{rcl} {\rm Ge} \ + \ 2 \, {\rm Br}_2 \ = \ {\rm GeBr}_4 \\ {\rm 72.6} & {\rm 319.7} & {\rm 392.3} \end{array}$ 

As shown in Fig. 234, a boat containing Ge powder, prepared by reduction of  $\text{GeO}_2$  at the lowest possible temperature, is heated in a combustion tube. The tube is connected by means of an adapter



Fig. 234. Preparation of germanium tetrabromide.

(sealed with a piece of rubber tubing) to a distilling flask serving as a receiver. The tube is connected on the other side to two wash flasks, the first with  $Br_2$  and the second with concentrated  $H_2SO_4$ . A stream of thoroughly dried  $CO_2$  is now passed through the apparatus. The reaction starts at a rather low temperature. At 200°C it becomes vigorous, and a crystalline mass, colored yellow by  $Br_2$ , collects in the cooling tube. From time to time it is melted down into the flask with a burner. This product is redistilled. It is not necessary to remove excess  $Br_2$  by treatment with Hg, since this can be accomplished without difficulty by simple distillation. In this way, a pure white product is obtained. The yield is almost quantitative.

 $\begin{array}{rcl} \text{GeO}_2 \ + \ 4 \ \text{HBr} \ = \ \text{GeBr}_4 \ + \ 2 \ \text{H}_2\text{O} \\ 104.6 & 323.7 & 392.3 & 36.0 \end{array}$ 

A tenfold excess of hydrobromic acid (d 1.78) is heated with 10.4 g. of  $\text{GeO}_2$  in a pressure flask. The heating is continued for 24 hours at  $180^{\circ}$ C in an oil bath. The  $\text{GeBr}_4$  formed is separated in a funnel and is distilled.

Additional preparative method: Finely pulverized Ge, reduced at a low temperature, is refluxed in a flask with an excess of  $Br_2$ (about 4 hours at 60°C), the excess  $Br_2$  is distilled off after removing the insoluble residue, and the crude product is fractionated.

PROPERTIES:

White, regular, glittering crystals (flattened octahedra). d (29°C) 3.123. B.p. 185.9°C (corr.), m.p. 26°C. Hydrolyzed by  $H_2O$ . Soluble in absolute alcohol, CCl<sub>4</sub>, benzene and ether.

**PROPERTIES:** 

- I. L. M. Dennis and F. E. Hance, Z. anorg. allg. Chem. <u>122</u>, 256 (1922); F. M. Brewer and L. M. Dennis, J. Phys. Chem. <u>31</u>, 1101 (1927).
- II. H. Bauer and K. Burschkies, Ber. dtsch. chem. Ges. <u>66</u>, 277 (1933); A. W. Laubengayer and P. L. Brandt, J. Amer. Chem. Soc. 54, 621 (1932).

### Germanium (IV) Iodide

GeL

 $\begin{array}{rrr} \mathrm{Ge} &+& 2\,\mathrm{I_2} = & \mathrm{GeI_4} \\ 72.6 & & 507.7 & & 580.3 \end{array}$ 

The preparation of  $\text{GeI}_4$  is similar to that of the tetrabromide. The necessary  $I_2$  vapor is produced simply by placing a boat with  $I_2$  in the front part of the tube in which the boat with the Ge powder is resting and heating the  $I_2$  boat as needed. When larger quantities of the  $I_2$  vapor are needed, they are produced in a retort, which is filled with  $I_2$  and through the opening of which  $CO_2$  is introduced. The neck of the retort is inserted into the combustion tube, and the joint is sealed with a piece of rubber tubing. The retort is heated in a water bath. The reaction begins at about 220°C and becomes vigorous after the tube reaches a temperature of 560°C. The  $\text{GeI}_4$ is purified in the same way as  $\text{GeBr}_4$ .

 $\begin{array}{rcl} \text{GeO}_2 \ + \ 4 \ \text{HI} \ = \ \text{GeI}_4 \ + \ 2 \ \text{H}_2\text{O} \\ 104.6 & 511.7 & 580.3 & 36.0 \end{array}$ 

After expelling the air with  $CO_2$ , 28 ml. of constant-boiling, colorless hydriodic acid is poured over 5 g. of  $GeO_2$  in a 100-ml. wide-neck round flask provided with a distilling head and an inlet tube for  $CO_2$ . The contents are slowly heated and kept boiling gently for 10 minutes. The  $GeO_2$  disappears and orange-red crystals are formed. The heating is increased and the water formed is distilled off. Finally the contents of the flask are sucked dry through a fritted glass filter crucible. Drying is continued in a desiccator without a vacuum. The material is purified by vacuum sublimation or, even better, by recrystallization from chloroform.

#### **REFERENCES:**

Orange-red, crystalline mass. Crystallizes in the cubic system. Soluble in carbon disulfide, chloroform and benzene, among other solvents. M.p.  $146^{\circ}$ C, b.p.  $\sim 350^{\circ}$ C; d ( $26^{\circ}$ C) 4.322.

**REFERENCES:** 

- I. L. M. Dennis and F. Hance, J. Amer. Chem. Soc. <u>44</u>, 2854 (1922); Z. anorg. allg. Chem. 129, 206 (1923).
- II. A. W. Laubengayer and P. L. Brandt, J. Amer. Chem. Soc. <u>54</u>, 621 (1932); L. S. Foster and A. F. Williston in W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London 1946, p. 112.

# Germanium Diodide

# Gel₂

A three-neck, 250-ml. flask with ground joints is filled with 20 g. of GeI<sub>4</sub>. The center neck is provided with a mercury-seal stirrer; the other two carry a reflux condenser and a ground glass stopper. Ten ml. of colorless, 57% hydriodic acid and 20 ml. of H<sub>2</sub>O are added, the stirrer is turned on, 7.6 ml. of 50% hypophosphorous acid is added (2 ml. excess), and the glass stopper is inserted. The contents of the flask are refluxed until the red crystals of GeI<sub>4</sub> are transformed into the yellow ones of GeI<sub>2</sub>. This happens in a short time. The material is cooled to 0°C; the crystals are filtered through a fritted glass filter crucible and washed with dilute hydriodic acid (2 parts H<sub>2</sub>O + 1 part 57% HI). The material is vacuum-dried over P<sub>2</sub>O<sub>5</sub> in a drying pistol heated with boiling toluene, whereby the remainder of the unreacted GeI<sub>4</sub> is sublimed off. Yield: 9 g. (75% of theoretical, based on GeI<sub>4</sub>).

 $\begin{array}{rl} \text{GeO} \ + \ 2 \ \text{HI} \ = \ \text{GeI}_2 \ + \ \text{H}_2\text{O} \\ 88.6 & 255.8 & 326.4 & 18.0 \end{array}$ 

Freshly precipitated GeO (see p. 711), prepared from 6 g. of GeO<sub>2</sub>, is filtered and washed and then reacted with 45 ml. of iodine-free, constant-boiling hydriodic acid. The resulting suspension is heated for half an hour at  $80^{\circ}$ C with stirring. After cooling, the precipitated crystals of GeI<sub>2</sub> are sucked dry and washed with 50 ml. of hydriodic acid diluted as in I. The drying procedure is the same as in I.

Other preparative method :  $GeS + 2 HI = GeI_2 + H_2S$ 

PROPERTIES:

Yellow crystals, CdI<sub>2</sub> structure type.

#### REFERENCES:

E. A. Flood, L. S. Foster and E. W. Pietrusza in W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London 1946, p. 106; L. S. Foster in L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London 1950, p. 63; H. M. Powell and F. M. Brewer, J. Chem. Soc. (London) 1938, 197.

### Trichlorogermane

GeS	+ 3 HCl	= GeHCl <sub>3</sub>	+	H₂S
104.7	109.4	180.0		34.1

The apparatus consists of a flask with ground joints and with inlet and outlet tubes. Several traps, connected in series, are attached to the outlet. The inlet is connected to a supply of HCl which was either passed over P<sub>2</sub>O<sub>5</sub> or prepared from solid NaCl and concentrated  $H_2SO_4$ . Vacuum-dried, precipitated GeS is added to the flask. The first trap is cooled with liquid nitrogen. The reaction starts spontaneously upon introduction of HCl. After completion of the run, the HCl supply is shut off, the apparatus is evacuated, and all volatile products of the reaction are condensed in the trap cooled with liquid nitrogen. The third trap is then cooled with liquid nitrogen, while the first is brought to -78°C. After the H<sub>2</sub>S and HCl are completely separated out, the second trap is cooled with liquid nitrogen and the first is brought to  $-45^{\circ}C$ (temperature of melting chlorobenzene). Two hours of standing in a vacuum suffices to remove the dissolved gases from the reaction product. The latter can then be fractionated under vacuum in the usual way and distilled into ampoules.

Additional preparative method: passing HCl over powdered Ge (with heating): Ge + 3 HCl = GeHCl<sub>3</sub> + H<sub>2</sub>.

#### **PROPERTIES:**

Colorless, mobile liquid. Distills at ~75°C. Loses HCl with the formation of GeCl<sub>2</sub> at about -30°C. The HCl remains dissolved in the GeHCl<sub>3</sub>. Therefore GeCl<sub>2</sub> can be prepared from GeHCl<sub>3</sub> at -30°C (under suction). Conversely, GeHCl<sub>3</sub> can be obtained from GeCl<sub>2</sub> by condensing HCl upon the latter. M.p. -71°C.

#### REFERENCES:

L. M. Dennis, W. R. Orndorff and D. L. Tabern. J. Phys. Chem. 30, 1049 (1926); C. W. Moulton and J. G. Miller, J. Amer. Chem. Soc. 78, 2702 (1956).

# Methylgermanium Triiodide

## CH<sub>3</sub>Ge I<sub>3</sub>

$$GeI_2 + CH_3I = CH_3GeI_3$$
  
326.4 142.0 468.4

The apparatus shown in Fig. 235 is connected to a vacuum pump. The thick-wall Pyrex ampoule a contains 10 g, of GeI<sub>2</sub>. After evacuation, 2.1 ml. (4.8 g.) of CH<sub>3</sub>I is allowed to distill from the graduated vessel b into the ampoule, which is cooled with acetone-Dry Ice mixture. The ampoule a is now sealed at the neck and heated for 24 hours at 110°C. (Caution: danger of explosion, particularly at the beginning, if the ampoule is too weak.) If the initial charge of Gel<sub>2</sub> is oxide-free, the crystals disappear as the reaction proceeds and a clear solution is obtained. Provided too great an excess of CH<sub>2</sub>I was not charged in, the yellow solution hardens on cooling. The excess CH<sub>a</sub>I is distilled off by cooling the side bulb c in a cooling mixture. The vessel is then opened; the



Fig. 235. Preparation of methyl germanium triiodide.

contents are placed in a suitable distillation apparatus and, after removing all remaining CH<sub>a</sub>I, finally distilled.

PROPERTIES:

Lemon-yellow rhombic crystals. M.p. 48.5°C, b.p. (752 mm.) 237°C. With rising temperature the crystals become reddish. Soluble in H<sub>2</sub>O (with hydrolysis) and in organic solvents, such as petroleum ether.

REFERENCES:

E. A. Flood, K. I. Godfrey and L. S. Foster in L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London 1950, p. 64; E. A. Flood, J. Amer. Chem. Soc. 55, 4935 (1933).

### Germanium Nitride

Ge<sub>3</sub>N<sub>4</sub>

$$3 \text{ Ge} + 4 \text{ NH}_3 = \text{Ge}_3 \text{N}_4 + 6 \text{H}_2$$
  
217.8 68.1 273.8 12.1

Powdered Ge, freshly reduced in a hydrogen stream at 600°C, is placed in a boat of sintered corundum or quartz. The boat is placed in an electrically heated quartz or porcelain tube, the temperature of which can be measured with a thermocouple. A stream of  $NH_3$  is passed over the boat. The reaction starts at about  $650^{\circ}C$ . The temperature must be maintained at about  $700^{\circ}C$ , since the nitride redecomposes into its elements above  $850^{\circ}C$ .

Additional preparative methods: a. Heating GeO<sub>2</sub> in a stream of NH<sub>3</sub> at 750°C: 3 GeO<sub>2</sub> +4 NH<sub>3</sub> = Ge<sub>3</sub>N<sub>4</sub> +6 H<sub>2</sub>O. b. Thermal decomposition of Ge(NH)<sub>2</sub>. The latter is obtained from GeCl<sub>4</sub> by ammonolysis according to the equation GeCl<sub>4</sub> +6 NH<sub>3</sub> = Ge(NH)<sub>2</sub> + 4 NH<sub>4</sub>Cl. The Ge(NH)<sub>2</sub> yields Ge<sub>3</sub>N<sub>4</sub> above 300°C, the decomposition going through the Ge<sub>2</sub>N<sub>3</sub>H stage.

#### PROPERTIES:

Colorless powder when pure; however, usually brownish. Crystallizes in a phenacite-type structure.

#### **REFERENCES:**

R. Schwarz and P. W. Schenk, Ber. dtsch. chem. Ges. <u>63</u>, 296 (1930); W. C. Johnson, J. Amer. Chem. Soc. <u>52</u>, 5160 (1930); R. Juza and H. Hah, Naturwiss. <u>27</u>, 32 (1939); R. Juza and A. Rabenau, Z. anorg. allg. Chem. <u>285</u>, 212 (1956); H. Hahn and R. Juza, Z. anorg. allg. Chem. <u>244</u>, 111 (1940).

### Germanium Disulfide

GeS<sub>2</sub>

A solution of  $\text{GeO}_2$  in 6N HCl is prepared. Alternatively, the solutions remaining from the hydrolysis of  $\text{GeCl}_4$  may be used, following removal of  $\text{Cl}_2$  with  $\text{H}_2\text{SO}_3$ . A rapid stream of  $\text{H}_2\text{S}$  is then introduced. The flask is tightly closed with a rubber stopper and is allowed to stand (preferably overnight) under the pressure of a Kipp apparatus. Pure white  $\text{GeS}_2$  precipitates out and is filtered off, washed successively with dilute HCl, alcohol and ether, and dried in a vacuum desiccator. The yield is quantitative. May be obtained in crystalline form by sublimation of GeS (at red heat) in a stream of sulfur vapor.

Additional preparative method: Passage of S-containing  $H_2S$  over GeO<sub>2</sub> at red heat.

PROPERTIES:

Precipitated: white, heavily chalking powder. Difficult to wet with water. Crystalline: flakes with a mother-of-pearl luster. d (14°C) 2.942. At about 800°C, melts to a dark liquid which hardens to an amber-yellow, transparent mass. Soluble in alkali as a thio salt. Can be converted to GeO<sub>2</sub> by dissolving in ammonia and oxidizing with  $H_2O_2$ . Volatile in an inert atmosphere at 800°C.

REFERENCE:

W. Pugh, J. Chem. Soc. (London) 1930, 2370.

### Germanium Monosulfide

GeS

A boat containing  $GeO_2$  is heated in a combustion tube while a gaseous mixture of  $H_2S$  and  $H_2$  is passed over it. The reaction begins at 500°C and gives an almost quantitative yield of the sulfide. A few hours are needed for the conversion of 10 g. of  $GeO_2$ . The GeS formed sublimes and is removed from the tube, pulverized and digested with cold, dilute ammonia. The byproduct  $GeS_2$  is thereby dissolved and solid GeS remains.

 $\begin{array}{rcrcrc} {\rm GeCl}_2 \ + \ {\rm H}_2{\rm S} \ = \ {\rm GeS} \ + \ 2 \ {\rm HCl} \\ {\rm 143.5} & {\rm 34.1} & {\rm 104.7} & {\rm 72.9} \end{array}$ 

The necessary GeCl<sub>2</sub> solution is prepared by reducing a solution of GeCl<sub>4</sub> with  $H_3PO_2$  under a CO<sub>2</sub> blanket, as described on page 711 for GeO. The reduction is complete when a 5-ml. sample shows at most a clouding, but no precipitate of GeS<sub>2</sub> on addition of 150 ml. of 6N  $H_2SO_4$  and introduction of  $H_2S$ . After reduction, the solution is cooled, and concentrated ammonia is added until a permanent precipitate of GeO appears. The solution is then saturated with  $H_2S$  under pressure and left to stand under  $H_2S$  for about 1 hour, with frequent shaking. The GeS precipitate is filtered and washed with water which has been weakly acidified with HCl. Drying is carried out under vacuum over  $P_2O_5$ .

**PROPERTIES:** 

If prepared according to method I, dark, gray-black crystals in reflected light, red to yellow-red in transmitted light. Red-brown crystals when prepared according to method II. M.p.  $\sim 530^{\circ}$ C, d 4.012. Soluble in HCl, reacts with HCl gas at room temperature according to the equation GeS + 2 HCl = GeCl<sub>2</sub> + H<sub>2</sub>S. Dry GeS is stable in air.

**REFERENCES:** 

W. Pugh, J. Chem. Soc. (London) <u>1930</u>, 2371; L. M. Dennis and R. E. Hulse, J. Amer. Chem. Soc. <u>52</u>, <u>3553</u> (1930); L. S. Foster in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London 1946, p. 102.

#### Tetraethoxygermane

 $Ge(OC_2H_5)_4$ 

Since  $GeCl_4$  does not react, as does  $SiCl_4$ , with alcohol to split off HCl, another procedure must be used:

A solution of 9.8 g. of metallic sodium in a considerable excess of absolute alcohol (the latter is distilled from Ca chips) is prepared. A solution of 20 g. of  $\text{GeCl}_4$  in absolute ether is added with stirring and exclusion of moisture. The mixture is refluxed for several hours and then decanted from the NaCl which separates. The solid is washed with absolute ether. The washings are combined with the main solution and the solvent is then driven off in vacuum. Finally, the tetraethoxygermane is distilled. It is redistilled at atmospheric pressure.

#### **PROPERTIES:**

Water-clear liquid. B.p. 186°C. Hydrolyzes easily, yielding GeO<sub>2</sub>.

#### **REFERENCES:**

D. L. Tabern, W. R. Orndorff and L. M. Dennis, J. Amer. Chem. Soc. <u>47</u>, 2043 (1925); R. Schwarz, P. W. Schenk and H. Giese, Ber. dtsch. chem. Ges. 64, 366 (1931).

### Germanium Tetraacetate

#### Ge(CH<sub>3</sub>COO)<sub>4</sub>

 $\begin{array}{rrrr} GeCl_4 & + & 4 \ Tl(CH_3COO) & = & Ge(CH_3COO)_4 & + & 4 \ TlCl \\ 214.4 & & 1053.7 & & 308.8 & & 959.4 \end{array}$ 

The apparatus consists of a three-neck, ground joint, 150-ml. flask, fitted with a dropping funnel, a mercury-seal stirrer and a reflux condenser equipped with a drying tube. A suspension of 50 g. of thallium acetate in 100 ml. of acetic anhydride is placed in the flask and a solution of 10 g. of GeCl<sub>4</sub> in 20 ml. of the anhydride is added dropwise from the funnel with vigorous stirring. Stirring is continued for 15 minutes on an  $80^{\circ}$ C oil bath and then at room temperature for another 45 minutes. The resulting TlCl precipitate is filtered off with exclusion of moisture and the filtrate is evaporated to 10 ml. under a vacuum of 20 mm. On cooling, the germanium tetraacetate separates out almost quantitatively in the form of fine needles. These are washed with acetic anhydride and anhydrous ether on a fritted glass filter crucible and dried in a vacuum.

**PROPERTIES:** 

Fine white needles. M.p.  $156^{\circ}$ C (decomposes before melting if heated too slowly). Hydrolyzes in water to acetic acid and GeO<sub>2</sub>. Soluble in acetic anhydride, benzene and acetone, only slightly soluble in CCl<sub>4</sub>.

**REFERENCE:** 

G. Schmidt, C. Blohm and G. Jander, Angew. Chem. 59, 235 (1947).

# SECTION 13

Tin and Lead

M. BAUDLER

Tin

Sn

#### TIN POWDER

Somewhat below its melting point  $(232^{\circ}C)$  tin becomes brittle, so that it can be broken down to apowder. A polymorphic  $\gamma$ -modification, to which this property was formerly ascribed, is not present according to recent x-ray studies.

Powdered tin can be commercially obtained in various grades of fineness. To prepare smaller quantities in the laboratory, pure tin is melted in a porcelain dish and heating is continued until the mass glows dark red. Liquid tin is poured off from the surface oxide scum into a preheated porcelain mortar and is vigorously pounded immediately after hardening. Since the metal cools down quickly to temperatures at which brittleness disappears, the mortar should be kept at about 200°C. If this is not done, the larger unbroken pieces must be heat treated again.

Because of its high reactivity, tin powder is used in place of granulated tin or tin bars for the preparation of various Sn compounds.

REFERENCES:

- L. Vanin. Handb. d. präp. Chem., 3rd Ed., Vol. I, Stuttgart, 1925. p. 587.
- C. L. Mantell. Tin, 2nd Ed., New York, 1949.

 $GRAY \alpha$  -TIN

Common, tetragonal  $\beta$ -Sn converts into powdery, gray  $\alpha$ -Sn on intense cooling to below 13.2°C, particularly in the presence of suitable catalysts.

Tin filings are mixed with a 10% solution of  $(NH_4)_2 SnCl_6$  in absolute alcohol in a closed flask and, when possible, some  $\alpha$ -Sn (about 1/100 of the Sn charge) is added. Complete conversion to gray  $\alpha$ -Sn occurs on standing for several days in a refrigerator at  $-5^{\circ}C$ . The recovered powder is carefully washed with cold, dilute hydrochloric acid, then with alcohol and ether, and dried in a vacuum desiccator at low temperatures.

**REFERENCES:** 

E. Cohen and C. van Eijk. Z. physik. Chem. <u>30</u>, 601 (1899).
E. Cohen. Z. physik. Chem. <u>35</u>, 588 (1900).

# Tin (II) Chloride

### SnCl<sub>2</sub>

Anhydrous tin (II) chloride is obtained by dehydration of the dihydrate:

 $SnCl_2 \cdot 2 H_2O + 2 (CH_3CO)_2O = SnCl_2 + 4 CH_3COOH$ 225.7 204.2 189.6

Commercial, crystallized  $SnCl_2 \cdot 2 H_2O$  (226 g.) is added with stirring to 204 g. of acetic anhydride (99-100%) in a 600-ml. beaker. The dehydration starts instantaneously with intense heat evolution and the  $(CH_3CO)_2O$  occasionally reaches boiling (use a hood); the anhydrous salt separates out in fine, white crystals at the same time. After 1.5 hours of this treatment it is filtered to dryness under suction, washed with two 15-ml. portions of dry ether, and dried in a vacuum desiccator. The yield is quantitative (189 g.). The product can be further purified by high vacuum distillation in a Vycor container or, preferably, in one of quartz.

Store in tightly closed ground glass bottles.

PROPERTIES:

White crystalline substance with an oily sheen; relatively stable in air; partial decomposition with hydrolysis and oxidation on continued exposure.

M.p. 247°C; the melt tends to supercool. B.p. 606°C; d 3.95. Rhombic crystals.

Readily soluble in water; the dihydrate crystallizes from concentrated solution; at greater dilutions, hydrolyzes with formation of Sn(OH)Cl; quite soluble in acetone, amyl alcohol, ethyl acetate, and absolute methanol and ethanol.

#### **REFERENCES:**

Organic Syntheses, Vol. 23, New York, 1943, p. 63. H. Stephen. J. Chem. Soc. (London) <u>1930</u>, 2786. W. Fischer and R. Gewehr. Z. anorg. allg. Chem. <u>242</u>, 188 (1939).

# Tin (IV) Chloride

SnCl₄

I.

 $Sn + 2 Cl_2 = SnCl_4$ 118.7 141.8 260.5

The apparatus shown in Fig. 236 is used, with vessel size chosen according to the amount of SnCl<sub>4</sub> desired. The Pyrex reaction vessel a is connected through the overflow tube e to the closed receiver c. The diameter of tube e must be large enough so that the liquid overflowing from a does not block the escape of gas from c into d. Tube d is directly connected (via a CaCl<sub>2</sub> drying tube) to the stack part of the hood arrangement. Flask b serves as a safety trap in case of liquid backup. Vessel  $\alpha$  is filled with pure Sn granules to a height of 1-2 cm. below the branching-off point of e. If possible, a few milliliters of SnCl<sub>4</sub> are added (to facilitate initiation of the reaction) so that the gas inlet tube f just barely dips below the surface of the liquid. A fast stream of pure, dry Cl<sub>2</sub> (see p. 272) is introduced through tube g. Immediately following the start of the reaction, a is cooled by immersion in water and the Cl<sub>2</sub> stream is so regulated that the SnCl<sub>4</sub> formed does not boil. An occasional glowing of the Sn beneath the liquid causes no harm. The vessel a fills rapidly with SnCl<sub>4</sub>, which then overflows into c. The



Fig. 236. Preparation of tin (IV) chloride.

yield is almost quantitative. Several kilograms of  $SnCl_4$  can thus be easily prepared within a day. The crude product from c and a is left standing for some time over Sn foil (with occasional shaking) to remove the dissolved  $Cl_2$ . The crude is then separated from excess metal by distillation in a ground glass apparatus, with careful exclusion of moisture; the boiling point of the pure substance is  $114^{\circ}C$ .

II. 
$$SnCl_4 \cdot 5 H_2O + 5 SOCl_2 = SnCl_4 + 5 SO_2 + 10 HCl_{350.6} 594.9 260.5 320.3 364.7$$

Somewhat more than the theoretically necessary amount of  $SOCl_2$  (see p. 382) is poured over crystallized  $SnCl_4 \cdot 5 H_2O$  in a flask with ground glass joints and the mixture is refluxed for a few hours, with a drying tube attached to the condenser. After the reaction, most of the unconverted  $SOCl_2$  is distilled off on a water bath, using the appropriate ground glass attachment. The last traces of  $SOCl_2$ , as well as of dissolved  $SO_2$  and HCl, are removed by repeated evacuation of the reaction flask at room temperature. The crude product is purified by careful fractionation in a column, while preventing access of moisture. The completely colorless middle fraction, boiling at  $114^{\circ}C$ , is collected.

### PROPERTIES:

Colorless liquid, fuming in air; takes up moisture, forming various hydrates and therefore is stable only when kept in hermetically closed vessels.

M.p. -30.2°C, b.p. +114°C; d. (20°C) 2.23.

Dissolves exothermically in water with extensive hydrolysis to colloidal stannic acid; miscible in all proportions with CS<sub>2</sub>.

**REFERENCES:** 

- I. H. Danneel, Angew. Chem. <u>39</u>, 1553 (1926); see also R. Lorenz, Z. anorg. Chem. 10, 44 (1895).
- II. H. Hecht, Z. anorg. Chem. 254, 37 (1947).

# Hexachlorostannic Acid

# $H_2SnCl_6 \cdot 6 H_2O$

$$\begin{array}{l} {\rm SnCl}_4 + 2\,{\rm HCl} + 6\,{\rm H}_2{\rm O} = {\rm H}_2{\rm SnCl}_6\cdot 6\,{\rm H}_2{\rm O} \\ {\rm 260.5} & 72.9 & 108.1 & 441.6 \end{array}$$

The stoichiometric quantity of  $H_2O$ ; in the form of concentrated hydrochloric acid (66.1 g., d 1.19) is added to 100 g. of pure SnCl<sub>4</sub> (see previous section). Considerable heat and HCl evolution accompany the reaction. When the reaction subsides, the flask is closed

with a two-hole rubber stopper with gas inlet and outlet tubes, both of which reach to the bottom of the flask. Pure, dry HCl (see p. 280) is introduced at room temperature. Absorption of the gas is favored by gentle rotation of the flask. At saturation (weight increase of about 8 g.), the gas flow is stopped and the reaction mixture is cooled by placing the flask in cold water. Crystals begin to settle out after a short time, and soon the entire contents of the flask solidify to a flaky, colorless mass. If pure materials have been used as reactants, the product is quite pure at this state. It can be separated into fractions of various grades of purity by repeated melting, partial crystallization and decantation of the mother liquor. The yield is almost quantitative.

Remains stable only when stored in sealed ampoules.

PROPERTIES:

Colorless, flaky crystals; very deliquescent in moist air with simultaneous liberation of HCl and formation of  $SnCl_4 \cdot 5 H_2O$ .

M.p. 19.2°C; gives off HCl at somewhat higher temperatures; d (27-28°C) 1.925.

REFERENCE:

K. Seubert. Ber. dtsch. chem. Ges. 20, 793 (1887).

# Ammonium Hexachlorostannate;

# Potassium Hexachlorostannate

(NH<sub>4</sub>)<sub>2</sub>SnCl<sub>6</sub>; K<sub>2</sub>SnCl<sub>6</sub>

 $\begin{aligned} & \text{SnCl}_4 + 2 \text{ NH}_4\text{Cl} = (\text{NH}_4)_2\text{SnCl}_6 \\ & 260.5 \quad 107.0 \quad 367.5 \\ & \text{SnCl}_4 + 2 \text{ KCl} = \text{K}_2\text{SnCl}_6 \\ & 260.5 \quad 149.1 \quad 409.6 \end{aligned}$ 

Pure, anhydrous  $SnCl_4$  (see p. 729) is dissolved in an equal weight of  $H_2O$  and is decomposed with a 50% excess of saturated M(I)Cl solution. White crystals of  $M(I)_2SnCl_6$  separate out on cooling. If desired, the volume of solution can be somewhat reduced beforehand by evaporation. To complete the crystallization, the solution is allowed to stand for some time in ice and the crystals are then sucked dry while cold. They are washed with some ice water and dried on clay in a desiccator. A concentrated solution of  $SnCl_4 \cdot 5 H_2O$ , containing some hydrochloric acid, may also be used in this reaction.

SYNONYM:

The ammonium salt  $(NH_4)_2 SnCl_6$ , is known in industry as "pink salt." It is used as a mordant in dyeing.
**PROPERTIES:** 

White, crystalline substances, may be stored in air. Both materials are very soluble in water; boiling causes precipitation of  $\text{SnO}_2 \cdot \text{nH}_2\text{O}$  from dilute solutions.

 $(NH_4)_2SnCl_6$ : d 2.39 K<sub>2</sub>SnCl<sub>6</sub>: d 2.71 Crystals have a J 1<sub>1</sub>-type structure (K<sub>2</sub>PtCl<sub>6</sub>).

REFERENCES:

H. F. Walton. Inorganic Preparations, New York, 1948, p. 110.
R. G. Dickinson. J. Amer. Chem. Soc. <u>44</u>, 276 (1922).
Bolley. Liebigs Ann. Chem. <u>39</u>, 100 (1841).

## Tin (II) Bromide

#### SnBr<sub>2</sub>

This compound is prepared by dissolving Sn in concentrated hydrobromic acid and concentrating this solution, thus forming  $H_2O$ - and HBr-containing crystals of SnBr<sub>2</sub>. Pure SnBr<sub>2</sub> is formed on further heating.

Concentrated hydrobromic acid is poured over Sn powder (see p. 727) contained in a flask, and the latter is warmed on a sand bath until the initial vigorous hydrogen evolution slows down. The liquid is then decanted from the undissolved Sn and evaporated on a water bath in the presence of some Sn foil until a salt scum forms on the surface. Needle-shaped crystals separate out on cooling. These are quickly sucked dry through a fritted glass filter and are placed in a vacuum over  $H_2SO_4$ , where they are surface-dried for a short time on clay. While still moist, the crystals are put into a round-bottomed Pyrex flask and heated in a stream of pure N2. An open flame is used. The flask is at first fanned carefully, then the flame is applied somewhat more directly. The water and the HBr adhering to the crystals come over first. With continued increase in temperature, larger amounts of HBr are liberated because of decomposition of the intermediate compounds present. Heating is continued until no further gas bubbles are produced and a clear melt of SnBr<sub>2</sub> is formed. After cooling in a stream of N<sub>2</sub>, the product may be purified by distillation in a quartz apparatus (under a nitrogen blanket). B.p. (1 atm.) in a weak stream of N<sub>2</sub>; 618°C.

#### PROPERTIES:

Formula weight 278.53. Bright yellow, crystalline substance; somewhat hygroscopic and sensitive to light.

M.p. 232°C, b.p.  $618^{\circ}$ C; partially oxidized to Sn (IV) compounds on heating in air; d 4.92.

Hydrolyzed by water but soluble without change inpyridine.

REFERENCES:

- F. Freyer and V. Meyer. Z. anorg. Chem. 2, 1 (1892).
- G. Wittig and H. Hartmann. Ber. dtsch. chem. Ges. 72, 1387 (1939).
- J. Kendall, E. D. Crittenden and H. K. Miller. J. Amer. Chem. Soc. <u>45</u>, 963 (1923).

## Tin (IV) Bromide

#### SnBr<sub>4</sub>

 $Sn + 2 Br_2 = SnBr_4$ 118.7 319.7 438.4

The reaction is carried out in a distilling flask with a long neck, the side arm of which is attached as close to the body of the flask as possible and ends in a CaCl, drying tube. Pieces of Sn, 2-3 cm, long, are placed in the flask, which is then closed above with a single-hole rubber stopper. A dropping funnel, with its tube drawn out to a capillary, is inserted in such a way that the tube ends in the lowest part of the neck of the flask but does not penetrate into the body of the flask. Pure Br<sub>2</sub> (see p. 275) is carefully added dropwise; this instantly produces a vigorous reaction, accompanied by a large heat evolution and possible ignition. Further addition of Br, must be so regulated that the reaction temperature stays below  $59^{\circ}$ C (b.p. of Br<sub>2</sub>) and no SnBr<sub>4</sub> or Br<sub>2</sub> penetrates into the side arm. When most of the tin is consumed and a sufficient amount of liquid collects on the bottom of the flask, the dropping funnel is replaced by a thermometer and the flask is manipulated so that the side arm points directly upwards. Excess Br<sub>2</sub> is removed by boiling for a few minutes, during which time the SnBr<sub>4</sub> condenses and runs back into the flask. When the product becomes nearly colorless, the flask is restored to normal position and the material is distilled into an attached receiving vessel, taking care to keep atmospheric moisture out. Freezing of the SnBr<sub>4</sub> to a snow-white crystalline mass is accompanied by a small increase in volume (caution, thin-walled receivers may burst). For further purification, the crude product may be remelted, partially solidified and the liquid poured off and discarded. Alternately, the last step may be replaced by fractional distillation in a ground glass apparatus.

To preserve for extended periods of time, must be stored in sealed ampoules or in ground glass bottles with tightly fitting stoppers. SYNONYM:

Tin tetrabromide, stannic bromide.

PROPERTIES:

White, crystalline substances; fumes somewhat in damp air; hygroscopic.

M.p.  $33^{\circ}$ C, b.p.  $201^{\circ}$ C; sublimes readily in a sealed tube; stable even on strong heating; d ( $35^{\circ}$ C) 3.35.

Dissolves in water with complete hydrolysis; soluble without change in  $AsBr_3$ .

Rhombic crystals, D<sub>2</sub>h

REFERENCE:

R. Lorenz. Z. anorg. allg. Chem. 9, 365 (1895).

## Tin (II) Iodide

#### SnI<sub>2</sub>

 $Sn + 2 HI = SnI_2 + H_2$ 118.7 255.9 372.5 2.0

Pure tin is covered with an excess of concentrated hydriodic acid in a flask with ground joints. A reflux condenser is then attached and the flask contents boiled until the metal dissolves and red crystals of  $SnI_2$  begin to separate out. The copious product which precipitates on cooling is sucked through a fritted glass filter crucible and recrystallized from alcohol. The crystals are finally dried in a vacuum desiccator over  $P_2O_5$ . For a very pure preparation the material can be repurified by high-vacuum distillation in a quartz apparatus.

II.  $SnCl_2 + 2 KI = SnI_2 + 2 KCl$  $(2 H_2O)$ 225.7 332.0 372.5

An agitated, moderately concentrated aqueous solution of  $SnCl_{2}$  is rapidly mixed with a KI solution of about the same concentration containing half the stoichiometric amount of KI. If a larger quantity of KI is added, a yellow double salt coprecipitates with the red  $SnI_{2}$  and is difficult to separate. After filtering the  $SnI_{2}$ , additional product may be obtained from the mother liquor by addition of the second half of the stoichiometric amount of KI. This method yields a preparation which, in general, is less pure than that obtained via method I. Purification and drying of the product are as in I.

I.

SYNONYM:

Stannous iodide.

**PROPERTIES:** 

Red, crystalline substance. M.p. 320°C, b.p. 720°C; d 5.28.

Somewhat soluble in water (100 g. of solution contains 0.96 g. of  $SnI_2$  at 19.8°C, 1.72 g. at 49.5°C, 3.70 g. at 97.3°C); also soluble in warm CHCl<sub>3</sub>, CS<sub>2</sub> and C<sub>6</sub>H<sub>6</sub>.

Rhombic crystals.

**REFERENCES:** 

- I. W. Reinders and S. de Lange. Z. anorg. allg. Chem. <u>79</u>, 230 (1913); W. Fischer and R. Gewehr. Z. anorg. allg. Chem. <u>242</u>, 188 (1939).
- II. B. Köhnlein. Liebigs Ann. Chem. <u>225</u>, 171 (1884); G. Wagner. Anorganisch-präparatives Praktikum, Vienna, 1947, p. 85.

# Tin (IV) Iodide SnI₄

 $Sn + 2I_2 = SnI_4$ 118.7 507.7 626.4

Six parts by weight of very pure  $CS_2$  are poured over one part by weight of Sn powder in a round flask with a ground glass stopper, and four parts by weight of  $I_2$  are gradually added. The stopper must be inserted except during actual addition. If large amounts of reactions are used, the flask must be cooled with ice. The red-brown solution formed is then drawn off from the excess of Sn (keep moisture out) and evaporated to dryness using an aspirator vacuum. Water vapor from the aspirator must be prevented from reaching the product. The red SnI<sub>4</sub> obtained is analytically pure.

SYNONYM:

Tin tetraiodide, stannic iodide.

**PROPERTIES:** 

Orange-red, needle-shaped crystals. Moisture sensitive, hydrolyze completely in water; soluble without change in  $CS_2$ , CHCl<sub>3</sub> and CH<sub>2</sub>I<sub>2</sub>. M.p. 143.5°C, b.p. 340°C, d 4.46.

Crystals: D11 structure type.

**REFERENCES:** 

- R. Schneider. Pogg. Ann. 127, 624.
- H. Hecht. Präparative anorganische Chemie, Berlin-Göttingen-Heidelberg, 1951, p. 102.

## Tin (II) Oxide

#### SnO

The preparation involves conversion of tin (II) chloride to hydrated tin (II) oxide, which is then dehydrated to SnO by extended heating in an aqueous solution.

A solution of C.P.  $SnCl_2 \cdot 2 H_2O$  is prepared in the smallest possible quantity of hot concentrated hydrochloric acid, and a  $Na_2CO_3$  solution is added until the mixture gives an alkaline reaction with phenolphthalein paper (not litmus paper, since the latter gives a reaction at too small an excess of alkali). The  $Na_2CO_3$  solution must be added carefully and gradually since the mixture foams over quite readily. The white hydrated Sn (II) oxide which settles out is heated for 2-3 hours in the supernatant liquid (bath liquid: saturated solution of NaCl, b.p. 110°C), thus causing quantitative conversion to blue-black, metallically lustrous SnO. The product is purified by washing several times with distilled water and dried at 110°C. Yield: about 80%.

Pure SnO should yield a clear solution with hydrochloric acid, but a very slight opalescence is acceptable.

SYNONYM:

Stannous oxide.

PROPERTIES:

Formula weight 134.70. Blue-black, crystalline substance with metallic sheen. In air, is oxidized to  $\text{SnO}_2$  above 220°C; in inert gas, decomposes to Sn and  $\text{Sn}_3\text{O}_4$  at ~400°C. An SnO melt is stable and hardens (with simultaneous disproportionation into Sn and  $\text{Sn}_3\text{O}_4$ ) at 1040°C. Above 1000°C, SnO attacks silicate-containing materials, forming Sn (II) silicate; vessels made of sintered corundum may be used up to 1600°C. B.p. (estimated) 1700°C, d 6.32.

Insoluble in water; soluble in acids, yielding Sn (II) salts. Crystals: B 10 structure type.

**REFERENCES:** 

W. Fraenkel and K. Snipischski. Z. anorg. allg. Chem. <u>125</u>, 235 (1922).

- M. Straumanis and C. Strenk. Z. anorg. allg. Chem. 213, 301 (1933).
- C. G. Fink and C. L. Mantell. J. Phys. Chem. 32, 103 (1928).
- H. Spandau and E. J. Kohlmeyer. Z. anorg. Chem. 254, 65 (1947).

## Stannic Acids

#### SnO₂ · n H₂O

#### $\alpha$ -STANNIC ACID

#### I. PREPARATION BY HYDROLYSIS OF AN Sn(SO4)2 SOLUTION

Very pure tin (30g.) is dissolved in 20 ml. of boiling concentrated  $H_2SO_4$  (at 1.84) and the solution is kept boiling until the separated sulfur agglomerates and the liquid clears. After cooling, the mixture is filtered through a fritted glass filter crucible. Concentrated  $H_2SO_4$  is again added to the filtrate to make the total volume 200 ml. For the hydrolysis the  $Sn(SO_4)_2$  solution is rapidly and dropwise added to one liter of distilled water (good agitation is necessary). The temperature of the water is maintained at  $0-2^{\circ}C$  by external cooling. A slightly opalescent liquid is formed. This is diluted the next day with 5 liters of ice-cold distilled water. Milky white  $\alpha$ -stannic acid separates out. It is allowed to settle for a short time, and most of the sulfuric acid is removed by washing the precipitate with successively larger amounts of distilled water. This is continued until the wash water gives an almost neutral reaction to litmus. The precipitate is then transferred to a leaf filter and the washing is continued by repeated resuspension in water and filtration. Complete removal of the adsorbed sulfuric acid (negative BaCl, reaction in the wash water) is usually attained only after a four-week treatment. The  $\alpha$ -stannic acid can be dried in air at room temperature and then ground to powder. However, it partially converts to the  $\beta$ -form during this operation.

#### II. REACTION OF AN Me(I)<sub>2</sub>Sn(OH)<sub>6</sub> SOLUTION WITH ACID

An agitated solution of Na<sub>2</sub>Sn(OH)<sub>8</sub> (see section on hydroxy salts) is slowly allowed to react with dilute hydrochloric acid at room temperature until the mixture is almost neutral. The copious  $\alpha$ -stannic acid precipitate settling out is treated as in method I. Complete removal of adsorbed foreign ions, particularly Na<sup>+</sup>, presents greater difficulty here than in method I, where no alkali is used.

## III. REACTION OF AN SnCl<sub>4</sub> SOLUTION WITH AMMONIA

A small excess of ammonia is added to an aqueous solution of  $SnCl_4 \cdot 5 H_2O$ , which may be cleared by adding a few drops of concentrated hydrochloric acid. The gelatinous white precipitate

of  $\alpha$ -stannic acid is purified as in method I. However, the complete removal of adsorbed foreign ions is extremely difficult.

PROPERTIES:

White, gelatinous precipitate; after drying, a white, glassy substance; increased dehydration and conversion to  $\alpha$ -stannic acid on aging; this process is accelerated by heat.

Freshly prepared, moist  $\alpha$ -stannic acid is soluble in sulfuric, hydrochloric and nitric acids as well as in sodium hydroxide, forming the corresponding Sn (IV) salts.

X-ray analysis shows that freshly prepared  $\alpha$ -stannic acid is amorphous; on aging, the material increasingly shows a faded powder pattern with SnO<sub>2</sub> lines.

REFERENCES:

- I. A. Gutbier, G. F. Hüttig and H. Döbling. Ber. dtsch. chem. Ges. <u>59</u>, 1232 (1926); W. Mecklenburg. Z. anorg. allg. Chem. <u>74</u>, 207 (1912).
- II. E. Posnjak. J. Phys. Chem. 30, 1073 (1926).
- III. H. B. Weiser and W. O. Milligan. J. Phys. Chem. <u>36</u>, 3030 (1932).

β-STANNIC ACID

The compound is prepared by oxidation of metallic Sn with concentrated nitric acid.

An excess of concentrated nitric acid (d 1.41) is poured over pure, granulated Sn in a deep porcelain dish, and the mixture is heated on a water bath with repeated addition of fresh acid. After cooling, most of the nitric acid is decanted. The powdered precipitate of  $\beta$ -stannic acid is digested several times with cold distilled water and washed on a filter until the wash water shows no further reaction with diphenylamine (about 25 washings are required). The product may be dried in air at room temperature.

PROPERTIES:

White, microcrystalline powder. Loses water increasingly on heating and finally converts to  $SnO_2$  at red heat.

Insoluble in sulfuric acid (as opposed to  $\alpha$ -stannic acid), concentrated hydrochloric and nitric acids, as well as concentrated sodium hydroxide.

Gives the same powder pattern as SnO<sub>2</sub>.

**REFERENCES:** 

A. Kleinschmidt. Monatsh. Chem. 39, 149 (1918).

- E. Posnjak, J. Phys. Chem. 30, 1073 (1926).
- H. B. Weiser and W. O. Milligan. J. Phys. Chem. <u>36</u>, 3030 (1932).

## Sodium Orthostannate

## Na<sub>4</sub>SnO<sub>4</sub>

 $SnO_2 + 2 Na_2O = Na_4SnO_4$ 150.7 124.0 274.7

Since both the Na<sub>2</sub>O starting material and the Na<sub>4</sub>SnO<sub>4</sub> product are sensitive to  $CO_2$  and moisture in the air, the pretreatment of starting materials, the reaction itself and the handling of the product must conform to certain rules. An all-glass apparatus is used and the preparation proceeds either in a vacuum or under an inert gas blanket. For details concerning suitable apparatus and its handling, as well as the techniques of working with exclusion of air, see the quoted original literature and Part I, p. 53 ff.

Pure  $\text{SnO}_2$  is dried in a vacuum at  $500^{\circ}\text{C}$  and is ground and thoroughly mixed with the stoichiometric amount of  $\text{NaO}_2$  in a vacuum ball mill (described in Part I, p. 76). When the powder becomes so fine that it begins to adhere to the glass walls, the grinding is stopped and the mixture is transferred into a sintered magnesia boat, excluding air as mentioned above. The boat, in a protective silver tube placed in the heated zone of a high-melting glass tube (I.D. of 25 mm.), is heated in a vacuum for some time at  $500^{\circ}\text{C}$ . Complete conversion to  $\text{Na}_4\text{SnO}_4$  takes place. The hard product of the reaction is loosened with a chisel while maintaining a flow of purified gas and the exact composition (which varies somewhat with the length of heating time) is determined by analysis. Occasionally, the white salt is tinted green or brown by some Ag from the protective tube; however, this impurity is undetectable by analytical or x-ray techniques.

#### PROPERTIES:

White, hygroscopic crystalline substance; reacts vigorously with water, forming a slightly cloudy solution, which clears on addition of oxalic acid.

**REFERENCE:** 

E. Zintl and W. Morawietz. Z. anorg. allg. Chem. 236, 372 (1938).

## Tin (II) Sulfide

SnS

## (Crystalline)

Sn + S = SnS118.7 32.1 150.8

The preparation from the elements cannot be carried out in a single step, since at the high temperatures used a portion of the

sulfur vaporizes before it can react. For this reason an excess of S must be used and the first reaction product must be repeatedly treated with additional sulfur until the approximate composition SnS is attained. Final purification is carried out in a stream of  $H_{2^*}$ .

A porcelain tube, closed on one side, is filled with a few grams of sulfur and heated in a furnace to about 900°C. A mixture of Sn and double the stoichiometric quantity of sulfur is added in portions to the tube, which by this time is filled with S vapor. A somewhat nonuniform cake is formed. The upper layer contains large, flaky crystals of SnS, while the under side is richer in Sn. After cooling, the product is pulverized, mixed with the same quantity of sulfur as before, and replaced in the heated tube. This second treatment often yields a homogeneous cake with the approximate composition SnS. If this is not the case, the heating must be continued with or without the addition of S, depending on the analysis. For purification, the crude product is transferred into an open porcelain tube and sublimed from a boat in a stream of H<sub>2</sub> at bright red heat. Any excess sulfur which might be present is driven off at lower temperatures and the pure SnS vaporizes at high heat. It condenses near the boat in beautiful shinv crystals.

II.

 $SnO_2 + 2 KSCN = SnS + K_2S + 2 CO + N_2$ 150.7 194.4 150.8 110.3 56.0 28.0

Very pure, completely anhydrous KSCN is used as the starting material. It is obtained by the repeated recrystallization of the commercial product from boiling 96% ethanol and drying under vacuum at 100°C. About 10 parts of the salt are melted in a porcelain crucible until the appearance of a blue color ( $t \approx 450$ °C), and 1 part of pure SnO<sub>2</sub> is then gradually added. The reaction is accompanied by a vigorous evolution of gas, and one should wait before each new addition of SnO<sub>2</sub> until a clear, flowing melt is reestablished. The temperature should not exceed the initial value ( $\approx 450$ °C) to avoid conversion of the SnS product to K<sub>2</sub>SnS<sub>3</sub>. When all the SnO<sub>2</sub> has been added, the mixture is further heated for 15 minutes and is then allowed to cool gradually. On treatment with water, the gray melt yields a residue of crystalline, analytically pure SnS.

#### PROPERTIES:

Dark blue-gray, crystalline mass with a bluish metallic sheen. The crystals are soft, friable and give a colored streak.

M.p. 882°C; on hardening, the melt expands noticeably between 400 and 600°C so that thin-walled vessels may burst. B.p. (in inert gas)  $\sim 1230$ °C; heating in air to a high temperature results in oxidation to SnO<sub>2</sub>; d 5.08.

Almost insoluble in water  $(1.36 \cdot 10^{-6} \text{ g. in } 100 \text{ g. at } 18^{\circ}\text{C})$ ; soluble (with chemical change) in concentrated hydrochloric acid and yellow ammonium sulfide.

Crystals: B 29 structure type.

**REFERENCES:** 

- I. W. Biltz and W. Mecklenburg. Z. anorg. allg. Chem. <u>64</u>, 226 (1909); A. Ditte. Comptes Rendus Hebd. Séances Acad. Sci. <u>96</u>, 1790 (1883).
- II. J. Milbauer. Z. anorg. Chem. <u>42</u>, 433 (1904).

## Tin (IV) Sulfide

#### SnS<sub>2</sub>

#### (Crystalline)

 $\begin{array}{c} (\mathrm{NH_4Cl}) \\ \mathrm{Sn} + 2 \, \mathrm{S} = \mathrm{SnS_2} \\ \mathrm{118.7} \quad \mathrm{64.1} \quad \mathrm{182.8} \end{array}$ 

The tin necessary for the preparation is added in the form of a reactive Sn amalgam. The latter is prepared by gently heating 6.5 parts of Hg in an evaporating dish on a sand bath (use a hood) with gradual addition of 13 parts of Sn chips. When the conversion is complete, the product is allowed to cool and is coarsely ground with a glass rod directly after solidification. The mass is then further pulverized. The entire amalgam is then finely ground with 8 parts of flowers of sulfur and 6.8 parts of finely pulverized NH<sub>4</sub>Cl. The mixture is then transferred to a Hessian crucible, which may be loosely covered with a clay or porcelain lid. The reaction mixture is moderately heated (about  $400^{\circ}$ C) on a sand bath for some length of time, either under a good hood or, still better, in the open (caution: mercury vapor). The heating is continued until no further vapors are liberated. The temperature is then quickly raised to red heat. As soon as S vapors begin to evolve or the brown mass begins to blacken at some points, the heating is stopped and the crucible is slowly cooled without increasing the movement of air. Depending on the charge, the reaction will last from 3 to 4 hours. After cooling, the crucible is carefully broken and the well crystallized upper layer, which glistens like gold, is separated. The layers below the topmost also contain SnS<sub>2</sub>. However, it is present in less well developed crystals. The bottom layer often contains some pure S. If the initial charge is too large, decomposition of the product (black discoloration) due to overheating may occur at the bottom and on the walls because of the poor heat transfer, while the center charge may not react thoroughly. The best

crystallizing layer is purified by sublimation on the sand bath. The translucent flakes of  $SnS_2$  thus obtained have a beautiful gold sheen; the yield is about 50%.

This product, under the name of "mosaic gold," is used in the paint industry. As "tin-bronze," it is also employed for bronzing. Instructions for preparing products with somewhat brighter or redder color tones are given in the appended original literature.

PROPERTIES:

Flaky or scaly crystals with a high gold-yellow sheen, soft as talc; very stable in air. On heating, the color deepens reversibly to dark red, then to deep brown, above about  $600^{\circ}$ C decomposition to SnS and S takes place; d 4.5.

Insoluble in water and mineral acids; soluble (with chemical change) in aqua regia, alkali sulfide and ammonium sulfide solutions.

Crystals: C6 structure type.

**REFERENCES:** 

W. Obst. Farbe und Lack 1927, 57.

H. Hadert. Chemiker-Ztg. 50, 7 (1926).

J. Lagutt. Angew. Chem. 1897, 557.

## Sodium Metathiostannate

## $Na_2SnS_3 \cdot 8 H_2O$

 $\begin{array}{ll} Na_4 SnS_4 + SnS_2 = 2 \ Na_2 SnS_8 \\ (18 \ H_2 O) \\ 663.2 \ 182.8 \\ \end{array} \begin{array}{l} (8 \ H_2 O) \\ 810.0 \end{array}$ 

A solution of 50 g. of pure  $Na_4SnS_4 \cdot 18 H_2O$  (see the following preparation) in 1.5 liters of distilled water is prepared. This solution is titrated at the boiling point with 1N HCl until a sample of the liquid gives a yellow color with bromocresol green (pH 4-5). To complete the separation of SnS<sub>2</sub>, the mixture is boiled gently for 4 hours. After cooling, the precipitate is very carefully washed and decanted 10 times (this takes several days) and the excess liquid is removed in a leaf filter. The pure SnS<sub>2</sub> is dried at 120°C and finely pulverized. The brownish powder is then placed in a boiling solution of 50 g. of  $Na_4SnS_4 \cdot 18 H_2O$  in 100 ml. of distilled water, which contains a few Sn grains for protection against air oxidation. Heating is continued until the reaction (with dissolution of the SnS<sub>2</sub>) is completed. The mixture is then evaporated and the concentrated solution is left standing in a crystallizing dish over CaCl<sub>2</sub>. The colorless salt precipitate is separated from the mother liquor, washed with some ice-cold water, and purified by double recrystallization from water. Yield: 37 g. of analytically pure  $Na_2SnS_3 \cdot 8 H_2O$ .

PROPERTIES:

Formula weight 405.02. Colorless, prismatic crystals, which lose their water of crystallization in vacuum over  $P_2O_5$  or by heating to 200-250°C. Readily soluble in water; 100 g. of solution at 16°C contains 38.1 g. of Na<sub>2</sub>SnS<sub>3</sub> · 8 H<sub>2</sub>O.

**REFERENCE:** 

E. E. Jelley. J. Chem. Soc. (London) 1933, 1580.

## Sodium Terrathiostannate (IV)

#### $Na_4SnS_4 \cdot 18 H_2O$

 $\begin{array}{c} Na_2Sn(OH)_8 + 4\,Na_2S = Na_4SnS_4 + 6\,NaOH \\ & (9\,H_2O) \\ 266.7 & 960.8 & 663.2 \end{array}$ 

For the preparation, 100 g. of technical grade  $Na_2Sn(OH)_6$ (about 80% pure) and 250 g. of  $Na_2S \cdot 9H_2O$  are dissolved in 700 ml. of boiling distilled water. The solution, colored greenish-black by impurities contained in the stannate, is held for 3 hours at 90 to 100°C. Then 40 g. of finely pulverized MgO is added and the solution is heated for another 2-3 hours. The precipitated impurities are suction-filtered and the slightly yellow filtrate is concentrated on a water bath to about 300 ml. On standing, colorless  $Na_4SnS_4 \cdot 18 H_2O$  separates out. The salt is placed on a fritted glass filter, washed with some ice-cold water, and purified by double recrystallization from water. The product is then placed on a clay plate and dried for a short time in air. It then has the composition indicated by the formula. On longer drying, part of the water of crystallization is lost. Yield: 80 g. of  $Na_4SnS_4 \cdot 18 H_2O$ .

PROPERTIES:

Colorless, crystalline substance; loses part of its water of crystallization on long standing in air or in vacuum over  $P_2O_5$  or on heating to 200-270°C; the last two moles are given off only at

red heat, but simultaneous decomposition of the salt to SnS and Na  $_2\!\mathrm{S}_{\mathrm{x}}$  occurs.

Readily soluble in water (57.1 g. in 100 g. of solution at 18°C). Monoclinic crystals.

**REFERENCE:** 

E. E. Jelley. J. Chem. Soc. (London) 1933, 1580.

## Tin (IV) Sulfate

## $Sn(SO_4)_2 \cdot 2 H_2O$

The preparation starts from  $\alpha$ -stannic and sulfuric acids. Freshly precipitated  $\alpha$ -stannic acid (see p. 737) is dissolved in an excess of hot, dilute sulfuric acid, and the colorless solution is evaporated. White crystals of Sn(SO<sub>4</sub>)<sub>2</sub> · 2 H<sub>2</sub>O separate. These acquire a needle-shaped, platelike, or prismatic appearance with increasing acid concentration. After cooling, the salt is suction-filtered through a fritted glass crucible and left for some times in a desiccator, on clay and over P<sub>2</sub>O<sub>5</sub>, in order to free it of the mother liquor. The pure product must be stored in sealed ampoules, since it is very hygroscopic.

**PROPERTIES:** 

Formula weight 346.85. Colorless, crystalline substance; very hygroscopic. Hydrolyzes completely in water, with separation of  $\alpha$ -stannic acid. Readily soluble in dilute sulfuric acid.

REFERENCES:

A. Ditte. Comptes Rendus Hebd. Séances Acad. Sci. 104, 172 (1887).

## **Tetra methyltin**

## Sn(CH<sub>3</sub>)<sub>4</sub>

The smoothest conversion and the best yields are obtained in the Grignard reaction whereby  $SnCl_4$  is treated with a  $CH_3MgBr$  solution:

$$\frac{4 \text{ Mg} + 4 \text{ CH}_{3}\text{Br}}{97.3 379.8}$$
 (ether)  $\frac{4 \text{ CH}_{3}\text{MgBr}}{1000}$ 

$$\frac{4 \text{ CH}_{\$} \text{MgBr} + \frac{\text{SnCl}_{4}}{260.5}}{(\text{ether})} \frac{\text{Sn}(\text{CH}_{\$})_{4} + 4 \text{ Mg}(\text{Cl. Br})_{2}}{178.8}$$

744

The Grignard reagent (CH<sub>3</sub>MgBr) is prepared in a 1-liter, two-neck flask provided with a reflux condenser and a gas inlet tube reaching to the bottom. A CaCl<sub>2</sub> drying tube is attached to the end of the condenser to prevent access of atmospheric The flask is charged with 24.5 g. of Mg shavings moisture. (about 1 gram-atom) and 500 ml. of carefully dried ether. The reaction is initiated with about 1 g. of "activated" Mg turnings. These are prepared as follows: About 1 g. of Mg turnings and 0.5 g. of I<sub>2</sub> are carefully heated in a dry test tube over a free flame, until most of the iodine sublimes onto the cool part of the tube. After cooling in a descicator the turnings, which are covered with a brownish layer, are placed in the reaction flask. Pure CH<sub>2</sub>Br from a cylinder or from a cooled supply trap is slowly added to the liquid via the gas inlet tube. If the conversion to CH<sub>3</sub>MgBr does not start within 3 minutes (which can be recognized by the persistence of the iodine color) then the CH<sub>3</sub>Br flow is interrupted and the flask is carefully heated on a water bath to 60 to  $70^{\circ}$ C. A vessel filled with ice water must be on hand to permit rapid cooling if the reaction is too vigorous. Once the reaction starts, the addition of CH<sub>3</sub>Br is so regulated that the ether remains at a moderate boil. It is absolutely necessary in further processing that the Mg be completely dissolved. About 120 g. of CH<sub>3</sub>Br (about 1.25 moles) is normally needed to accomplish this, but considerably more may be if this reagent is introduced too rapidly. In the latter case, most of the methyl bromide escapes through the condenser without reacting. If necessary, the last traces of Mg can be converted by adding about 5 g. of CH<sub>a</sub>I through the reflux condenser. After the addition, the reaction mixture is refluxed for half an hour on the water bath. The flask is then closed off with a CaCl<sub>2</sub> tube and allowed to stand at room temperature until further use.

The reaction with  $SnCl_2$  is carried out under a hood, using a three-neck, 1-liter ground glass flask provided with a reflux condenser, a well-sealed stirrer and a dropping funnel. Both the condenser and the dropping funnel are equipped with  $CaCl_2$  drying tubes. The  $SnCl_4$  cannot be added directly to the Grignard reagent, as is usually done in analogous preparations. This reaction is too violent since even the reaction of  $SnCl_4$  with ether, which yields a crystalline etherate, is very exothermic. It is therefore much more practical to prepare this etherate separately and then add to it the Grignard solution.

The reaction flask is charged with 200 ml. of absolute ether, and 45 g. of anhydrous  $SnCl_4$  (see p. 729) is added dropwise, while vigorously stirring and cooling with ice water. After the addition, the dropping funnel is replaced with a clean one and the ethereal solution of  $CH_3MgBr$  is added over a period of 45 minutes with vigorous agitation. That addition proceeds at room temperature. The reaction is completed by refluxing for 2 days (twice for 10 hours) on the water bath. The product is then checked for malodorous methyltin halides which are initially present. If the odor is present, refluxing must be continued. The reaction mixture is then carefully decomposed with distilled water from the dropping funnel. The flask must be cooled with ice water, and addition is continued until the initial effervescence subsides. Finally, 10% hydrochloric acid is added until the precipitated Mg salt dissolves completely and two layers can be observed in the solution (if necessary, let stand for some time). The ether layer is separated in a separatory funnel and washed successively with some water and a 5% KF solution. Any methyltin halides still present are thus converted to the corresponding fluorides. These are insoluble and can be filtered off. The ether solution is dried for several hours with CaCl<sub>2</sub>. Then most of the solvent is removed in a slow distillation with a suitable column. The remainder is fractionated at atmospheric pressure, using the same column. The boiling point of the pure substance is 76°C. The yield corresponds to about 90% of theoretical, based on the SnCl, used.

#### PROPERTIES:

Water-clear, highly refractive, mobile liquid with an agreeable, sweet, ether odor; poisonous; stable to air and water. B.p. 76°C,  $d_4^{26}$  1.291. Insoluble in water; miscible in all proportions with ether, absolute alcohol and other organic solvents.

#### **REFERENCES:**

F. Ossenbrink. Thesis, Cologne, 1952; see Also E. Krause and A. von Grosse. Die Chemie der metall-organischen Verbindungen, Berlin, 1937, p. 314 ff.

## Tetraethyltin

## $Sn(C_2H_5)_4$

 $\begin{array}{c} 4\,C_{2}H_{5}Br + 4\,Mg = 4\,C_{2}H_{5}MgBr \\ 435.9 \qquad 97.3 \end{array}$ 

$$\begin{array}{l} 4\,C_2H_5MgBr + \,SnCl_4 = Sn(C_2H_5)_4 + 4\,Mg(Cl,Br)_2 \\ 260.5 \quad 234.9 \end{array}$$

The preparation is analogous to that of  $Sn(CH_3)_4$ . The reader is referred to the detailed description of the procedure given under that compound, unless changes are expressly indicated in what follows. An identical two-neck flask is used for the preparation of the  $C_2H_5MgBr$  solution. However the second neck carries a dropping funnel for the addition of  $C_2H_5Br$  instead of a gas inlet tube. About 136 g. of  $C_2H_5Br$  is needed for the conversion of 24.5 g. of Mg shavings.

Anhydrous  $SnCl_4$  (45 g.; see p. 729) is carefully added drop-wise to the absolute ether solution of  $C_2H_5MgBr$  in the apparatus previously described for the further reaction. This addition must be done under a hood. The reaction flask is cooled with flowing water. With larger charges the preparation of the SnCl<sub>4</sub> etherates should be carried out separately and the Grignard solution should then be added dropwise. After completion of the addition, the mixture is refluxed for one hour and the ether is then completely distilled off on a water bath. The residue is heated for 1/2 hour on a boiling water bath and after cooling is remixed with the ether previously removed. Finally, with the reflux condenser in place, water and 5% hydrochloric acid are carefully added from the dropping funnel until a clear separation of the layers is observed. The ether solution is then processed in the same way as  $Sn(CH_3)_4$ . Because of its high boiling point, the last fractionation of the very concentrated product is carried out under aspirator vacuum. Boiling point of the pure substance (13 mm.) is 78°C.

The yield is approximately 75%.

PROPERTIES:

Colorless, highly refractive, mobile liquid with an agreeable, sweet, ether odor; poisonous; stable to air and water; quite flammable.

M.p -112°C, b.p. 175°C; d<sup>25</sup><sub>4</sub> 1.192.

Insoluble in water; miscible in all proportions with ether, absolute alcohol and other organic solvents.

REFERENCES:

- E. Krause and A. von Grosse. Die Chemie der metall-organischen Verbindungen [The Chemistry of Organometallic Compounds] Berlin, 1937, p. 314 ff.;
- F. Ossenbrink. Thesis, Cologne, 1952.

## Tin (IV) Acetate

## Sn(CH<sub>3</sub>COO)<sub>4</sub>

 $\begin{array}{l} SnI_4 + 4 \ TlCH_3COO = Sn(CH_3COO)_4 + 4 \ TlI \\ 626.4 \ 1057.7 \ 354.9 \ 1329.2 \end{array}$ 

The reaction of  $TlCH_3COO$  with  $SnI_4$  requires exclusion of moisture. It proceeds in a 150-ml. three-neck ground glass flask

which is equipped with a mercury-seal stirrer, a straight, ground glass filling tube, and a reflux condenser with a drying tube. The flask is filled with a suspension of 16.8 g. of TICH, COO in 100 ml. of pure acetic anhydride and 10 g. of SnI, is added by portions with constant, vigorous stirring. The stopper on the filling tube should stay in place except for the actual short addition time. The reaction starts immediately and the yellow, sparingly soluble TII settles out. After the addition, vigorous stirring is continued for 1.5 hours at 80°C and for 0.5 hour at room temperature. The TII is then rapidly suction-filtered with exclusion of moisture. The filtrate is concentrated to about 50 ml. with moderate heating at 20 mm. (keep out moisture). On cooling, a precipitate of white needles of  $Sn(CH_3COO)_4$  is obtained. This is filtered in the absence of moisture, washed with anhydrous ether and finally dried in vacuum. Further concentration of the yellow-orange mother liquor vields an additional pale vellow fraction of the salt which may be further purified by recrystallization from acetic anhydride if necessary. The yield is practically quantitative.

PROPERTIES:

White, crystalline substance; very moisture sensitive. M.p.  $253^{\circ}$ C. Hydrolyzes in water to stannic and acetic acids; quite soluble in benzene and acetone, moderately soluble in CCl<sub>4</sub>.

REFERENCE:

H. Schmidt, C. Blohm and G. Jander. Angew. Chem. A59, 233 (1947).

#### Lead

#### Pb

#### LEAD, PUREST FORM

Commercial electrolytic lead (about 99.995%) is sufficiently pure for most laboratory purposes. However, it contains, depending on the method used in its production, varying concentrations of minute amounts of Cu, Bi, Fe, Zn, Cd, As, Sb, Sn, Se, Te and rare metals, totaling about  $5-50 \cdot 10^{-6}$  g. of impurities/g. Pb. No significant purity improvement would be obtained by additional electrolytic refining. Thus, the lead required for special investigations, where the highest purity is needed, must be purified by processes other than electrolytic. One such method consists of the following.

A solution of 10 kg. of C.P. Pb(CH<sub>3</sub>COO)<sub>2</sub> · 3 H<sub>2</sub>O in 5 kg. of C.P. glacial acetic acid and 500 g. of C.P. acetic anhydride is prepared by gentle heating of the mixture. After cooling to room temperature, 100 g. of thioacetic acid (pure) is quickly added, while good agitation is maintained. The mixture is then heated on a water bath until the flocculation of the PbS is complete. After standing, it is suction-filtered through fritted glass and the filtrate is diluted with enough double-distilled water so that most of the  $Pb(CH_3COO)_2 \cdot 3 H_2O$  is reprecipitated. The crystalline slurry is refiltered and the dilution procedure is repeated once or twice. Finally the purified salt is dried and portions of it heated in covered porcelain crucibles, whereby metallic Pb and a small amount of oxide are formed. Care must be taken to prevent melting of the PbO formed during the heating. Otherwise a lead silicate slag is formed on the wall of the crucible, from which silicic acid and possibly even molten metals can migrate into the molten Pb. Following the decomposition, the liquid metal is poured into a suitable porcelain vessel, while carefully avoiding contamination by the oxide scum. All vessels and apparatus used in this process must be thoroughly prerinsed with a brominesulfuric acid mixture and double-distilled water. In place of thioacetic acid, a solution of 100 g, of thiourea (C.P.) in hot absolute alcohol can be used for the precipitation.

The Pb thus obtained is practically free of Se and Te as well as of all those elements whose sulfides have a lower solubility product than PbS. In addition, all colloidal impurities present in Pb(CH<sub>3</sub>COO)<sub>2</sub> · 3 H<sub>2</sub>O (for example, Au) are removed. The noble metal content after three purification steps is  $10^{-12}$  g. Au and  $10^{-40}$  g. Ag/g. Pb. Very pure Pb is strikingly soft and easy to cut, and on melting is more resistant to air oxidation than the very pure commercial electrolytic material.

In testing for impurities, spectrographic analysis is applicable up to the order of magnitude of  $10^{-5}$  g./g. Pb; Russell recommends colorimetric methods for smaller traces; see also the analytical procedure of Hemingway.

**REFERENCES:** 

- A. E. van Arkel. Reine Metalle. [Pure Metals], Berlin, 1939, p. 503.
- R. C. Hughes. J. Electrochem. Soc. 101, 267 (1954).
- F. Haber and J. Jaenicke. Z. anorg. allg. Chem. <u>147</u>, 156 (1925).
- R. S. Russell. Proc. Australasian Inst. Mining Met. (N.S.) <u>87</u>, 167 (1932); <u>95</u>, Appendix I, 152 (1934); Hemingway, Proc. Australasian Inst. Mining Met. (N.S.) <u>47</u>, 245 (1922); Brit. Eng. Stand. Assoc., Stand. Spec. <u>1928</u>, No. 334; Amer. Soc. Test. Mat. Standards, Triennal Issue 1930, p. 789.

# Lead (IV) Chloride

## **PbCl₄**

 $\begin{array}{c} PbCl_2 + 2 \ HCl + Cl_2 + 2 \ C_5H_5N = (C_5H_5NH)_2PbCl_6 \\ 278.1 \quad 72.9 \quad 70.9 \quad 158.2 \quad 580.1 \end{array}$ 

 $(C_5H_5NH)_2PbCl_6 + H_2SO_4 = PbCl_4 + (C_5H_5NH)_2SO_4 + 2 HCl_{580.1} + 98.1 + 349.0 + 256.3 + 72.9$ 

To prepare pyridinium hexachloroplumbate (IV), a 600-ml. wash flask is filled with 20 g. of very finely pulverized PbCl<sub>2</sub> and 400 ml. of concentrated hydrochloric acid. A vigorous stream of Cl<sub>2</sub> (2-3 bubbles/sec.) is then introduced while the flask is frequently shaken. The lead salts dissolve completely within 2-3 hours (if greater amounts of the compound are desired, several wash flasks filled the same way may be connected in series). The contents of the flask are cooled to 0°C and the  $(C_5H_5NH)_2PbCl_6$  is precipitated by adding 7 g. of pyridine. An additional 3 g. of pyridine is added to the supernatant liquor. The bright yellow compound, which is suction-filtered, is then washed with about 50 ml. of 96% alcohol and dried at 50°C. Pyridinium hexachloroplumbate (IV) decomposes instantly in water, yielding a precipitate of PbO<sub>2</sub>.

To prepare PbCl<sub>4</sub>, 20 g. of (C<sub>5</sub>H<sub>5</sub>NH)<sub>2</sub>PbCl<sub>6</sub> is added with slow stirring to 600 g. of concentrated  $H_2SO_4$  (cooled to  $-10^{\circ}C$ ) over a period of 10 minutes. With slow, continuous stirring, the mixture is allowed to warm up to 0°C and is kept at this temperature for 1 hour. A shorter holding time is insufficient for complete separation of the PbCl<sub>4</sub>. A clear yellow oil settles to the bottom of the flasks, and the  $H_2SO_4$  shows a slight milky turbidity caused by the very fine precipitate. The acid is then decanted and the oil is quickly washed in a dry separatory funnel with 50 ml. of con-centrated  $H_2SO_4$ , cooled to  $-10^{\circ}C$ . The oil and acid mixture is shaken vigorously, and the pure oil, which settles in a short time, is allowed to flow into a receiver. Yield: 8 g. (66% of theoretical). Since the pure substance decomposes easily in air, especially at somewhat elevated temperatures, it is preferably kept in closed flasks under pure, concentrated H<sub>2</sub>SO<sub>4</sub> and stored in the dark at -80°C. The preparation of larger amounts is not entirely without danger, since under certain circumstances an explosive decomposition to PbCl<sub>2</sub> and Cl<sub>2</sub> may occur.

#### SYNONYM:

Lead tetrachloride, plumbic chloride.

**PROPERTIES:** 

Clear, yellow, highly refractive liquid; fumes in moist air; unstable, yields  $Cl_2$  and forms  $PbCl_2$  (observed as turbidity); may be stored for a time in the dark if kept under concentrated  $H_2SO_4$  and at low temperatures.

M.p.  $-15^{\circ}$ C. Heating accelerates the decomposition and leads under certain circumstances, to explosive decomposition; d (0°C) 3.18.

Hydrolyzes in water, yielding  $PbO_2$ ; soluble in anhydrous  $CHCl_3$  and  $CCl_4$  as well as in concentrated hydrochloric acid.

REFERENCE:

W. Biltz and E. Meinecke. Z. anorg. allg. Chem. 131, 1, (1923).

## Ammonium Hexachloroplumbate

## (NH<sub>4</sub>)<sub>2</sub>PbCl<sub>6</sub>

I.

 $PbCl_2 + Cl_2 + 2 HCl = H_2PbCl_6$ 278.1 70.9

 $\begin{array}{l} H_{2}PbCl_{6}+2\ NH_{4}Cl=(NH_{4})_{2}PbCl_{6}+2\ HCl\\ 107.0\ 456.0\end{array}$ 

A large porcelain mortar is used to grind 30 g. of pure PbCl<sub>2</sub> with 60 ml. of concentrated hydrochloric acid (d 1.19), and the resulting suspension is poured into a 1-liter, flat-bottomed flask. The residue of coarser particles remaining in the mortar is treated several times in the same way, each time using the same amount of acid, until all the PbCl<sub>2</sub> is transformed into a fine powder suspended in 600 ml. of concentrated hydrochloric acid. The flask is then cooled in an ice bath and a moderately fast stream of pure Cl<sub>2</sub> (see p. 272) is introduced. Absorption of the gas is facilitated by frequently rotating the flask. The liquid becomes yellow after a short time and the PbCl<sub>2</sub> dissolves in 1-2 hours, forming  $H_2PbCl_6$ . When most of the  $P\tilde{b}Cl_2$  has dissolved, the residue may sometimes react very slowly. If that is the case, one can either accelerate the oxidation by adding concentrated HCl and continuing the introduction of Cl<sub>2</sub>, or the loss of yield may be neglected and the suspension transferred to a fritted glass filter.

The clear, ice-cold  $H_2PbCl_6$  solution is then mixed with an ice-cold solution of 12 g. of  $NH_4Cl$  in 120 ml. of water and the mixture is left standing for several hours in ice. (If a more concentrated  $NH_4Cl$  solution is used, the product is frequently

contaminated with solid  $NH_4Cl$ .) A fine, yellow precipitate of  $(NH_4)_2PbCl_6$  is allowed to settle out and is rapidly filtered through an ice-cooled filter. The filter cake is washed with ice-cold absolute alcohol and ether until the filtrate is free of HCl and  $Cl_2$ . It is then dried on clay in a desiccator. Yield: about 35 g.

# II. REACTION OF NH<sub>4</sub>Cl WITH AN ELECTROCHEMICALLY PREPARED SOLUTION OF H<sub>2</sub>PbCl<sub>3</sub>

The electrochemical preparation of an  $H_2PbCl_e$  solution is based upon the electrolysis of hydrochloric acid with a Pb cathode and two anodes. One anode is made of lead and dissolves, yielding Pb<sup>2</sup>+ ions. The other anode, which is not attacked, is made of carbon. Further oxidation of the ions to Pb<sup>4</sup>+ takes place at the carbon electrode.

A clay cup is placed in the center of a battery jar (see Fig. 240) and serves as the cathode compartment. A lead plate (7 cm. long, 3.5 cm. wide) with a strap for lead connection and for support is placed in the jar. The bottom of the jar is covered with a plate of Acheson graphite  $(12.5 \times 7.5 = 94 \text{ cm}^2)$ , which serves as the unattacked anode. A carbon rod (1.5 cm. diameter) is screwed into one corner of the plate. This latter is surrounded by a somewhat larger glass tube, which extends from the base to above the surface of the liquid. Two corrugated Pb metal plates (each 27 cm. long and 5 cm. wide) are placed on either side of the clay cup to serve as dissolving anodes. The upper ends of these anodes are bent over the edge of the jar. All Pb electrodes are well cleaned with a wire brush prior to the run. The carbon rod and Pb anodes are connected in parallel. Since the current should be independently regulated in each anode loop, a rheostat and an ammeter should be included in each circuit. The anode compartment is filled with 1200 ml. of HCl with a density of 1.18, the cathode compartment with 225 ml. of HCl with a density of 1.10. Since the temperature of the anode electrolyte may not rise above 10°C during the experiment, the entire battery jar is cooled from the outside with ice water. Electrolysis proceeds at a potential of 12-14 volts and a current of 2 amp, in each of the two loops, so that the current density on the Pb anode is 0.005 amp./cm<sup>2</sup>, while that on the carbon electrode is 0.03 amp./cm? . The length of the run should be governed by the requirement of 20-25 ampere-hours per liter of anode fluid. The current efficiency, based on H<sub>2</sub>PbCl, is 70-80% under these conditions. If the electrolysis lasts too long, there is a marked decrease in efficiency since the H\_PbCl\_ becomes increasingly involved in the current passage, with consequent evolution of Cl, at the anode. Also, particles which separate from the badly corroded Pb anodes interfere with the electrolysis. The current

is shut off at the end of the run and the orange-yellow analyte is poured into a beaker.

Precipitation of the solution of  $H_2PbCl_6$  with an excess of 10%  $NH_4Cl$  solution and subsequent filtration and drying of the precipitate are carried out as in method I. Yield: about 65 g. of pure  $(NH_4)_2PbCl_6$ .

The  $(NH_4)_2PbCl_8$  is used as a raw material for the preparation of  $PbCl_4$ .

SYNONYMS:

Ammonium lead (IV) chloride, ammonium plumbic chloride.

PROPERTIES:

Lemon-yellow, crystalline powder; stable in air. Becomes orange-yellow at 70 to  $80^{\circ}$ C; decomposes above  $130^{\circ}$ C into Cl<sub>2</sub>, NH<sub>4</sub>Cl and PbCl<sub>2</sub>.

Hydrolytically decomposed by water, separating  $PbO_2$ ; soluble without change in 20% hydrochloric acid.

Crystals: J 1, structure type.

REFERENCES:

- I. H. Friedrich. Monatsh. Chem. <u>14</u>, 505 (1893); Ber. dtsch. chem. Ges. <u>26</u>, 1434 (1893); H. Hecht. Präparative Anorganische Chemie, Berlin, 1951, p. 151.
- II. K. Elbs and R. Nubling. Z. Elektrochem. 9, 776 (1903); E. Müller. Electrochemisches Praktikum, 7th Ed., Dresden and Leipzig, 1947, p. 225.

## Potassium Hexachloroplumbate

K<sub>2</sub>PbCl<sub>6</sub>

 $\begin{array}{rrrr} PbCl_2 \ + \ Cl_2 \ + \ 2 \ HCl \ = \ H_2PbCl_6 \\ 278.1 & 70.9 \end{array}$ 

$$H_{2}PbCl_{6} + 2 KCl = K_{2}PbCl_{6} + 2 HCl$$
  
149.1 498.1

A solution of  $H_2PbCl_g$  is prepared by introducing  $Cl_2$  into a suspension of 30 g. of PbCl<sub>2</sub> in 600 ml. of concentrated hydrochloric acid at room temperature [see the procedure under  $(NH_4)_2PbCl_g$ ]. When the solution becomes saturated with  $Cl_2$ , it is decanted from the unreacted PbCl<sub>2</sub> and rapidly cooled with ice. An ice-cold solution of 15 g. of KCl in 200 ml. of water is added to 500 ml. of the clear ice-cold  $H_2PbCl_6$  solution, and a stream of pure HCl gas is introduced into the mixture, which is held at 0°C (see also the description of the preparation of ammonium hexachlorotitanate in the section on Titanium). Separation of the lemon-yellow  $K_2PbCl_6$  begins after a short time, and it is completed by further introduction of HCl until the solution is saturated at 0°C. As soon as the precipitation is complete, the finely crystallized salt is suction-filtered on fritted glass, washed with some cold concentrated hydrochloric acid and dried by pressing on a clay plate in air.

It is essential that the solution of  $H_2PbCl_6$  be used immediately after its preparation. Otherwise some decomposition occurs and PbCl<sub>2</sub> is formed. In the presence of the latter an orange-brown, monoclinic product of unknown constitution precipitates out instead of the yellow, cubic salt.

SYNONYM:

Potassium lead (IV) chloride, potassium plumbic chloride.

**PROPERTIES:** 

Lemon-yellow, crystalline powder, stable in air for several days; however, it gradually decomposes with fading of the yellow color (hydrolysis). Decomposes at higher temperatures, evolving  $Cl_{2}$ .

Hydrolyzes in water, forming PbO<sub>2</sub>. Soluble without decomposition in 20% hydrochloric acid.

Crystal: J 1, structure type.

**REFERENCES:** 

H. L. Wells. Z. anorg. allg. Chem. 4, 335 (1893).

K. Elbs and R. Nübling. Z. Elektrochem. 9, 776 (1903).

A. Gutbier and M. Wissmüller. J. prakt. Chem. <u>90</u>, 491 (1914).

G. Engel. Z. Kristallogr. 90, 341 (1935).

H. Leibiger. Thesis, Univ. of Freiburg in Breisgau, 1951.

## Potassium Iodoplumbite

## KPbI<sub>3</sub>·2H<sub>2</sub>O

 $\begin{array}{rrrr} Pb(NO_3)_2 &+ 3\ KI &= \ KPbI_3 &+ 2\ KNO_3 \\ & & (2\ H_2O) \\ 331.2 & 498.1 & 663.1 & 202.2 \end{array}$ 

A solution of 4 g. of  $Pb(NO_3)_2$  in 15 ml. of warm, distilled water is prepared and mixed with a warm solution of 15 g. of

KI in 15 ml. of distilled water. Good agitation is necessary. Yellow  $PbI_2$  precipitates out, and on cooling gradually transforms to pale yellow  $KPbI_3 \cdot 2 H_2O$ . On renewed heating, the yellow color of  $PbI_2$  reappears because of the strong secondary dissociation of the complex. The salt is suction-dried in a fritted glass filter and finally dried by pressing between pieces of filter paper or on a clay plate.

The anhydrous compound can be prepared from the dihydrate either by storing the latter for a period of time in a vacuum desiccator over concentrated sulfuric acid or by dissolving it in 15-20 ml. of acetone and then precipitating with a double volume of ether.

Anhydrous  $KPbI_3$  is a sensitive moisture indicator since it produces the yellow  $PbI_2$  rather than the hydrate. See the article by Biltz for particulars of detection of traces of  $H_2O$  in gases or organic solvents with either solid  $KPbI_3$  or  $KPbI_4$  dissolved in acetone.

SYNONYMS:

Potassium lead (II) iodide, potassium plumbous iodide.

**PROPERTIES:** 

The dihydrate forms pale yellow, needle-shaped crystals, stable in air. On more rigorous drying in a desiccator, the water of crystallization is given off with the formation of whitish, powdery  $\text{KPbI}_3$ , which immediately turns yellow in moist air (formation of PbI<sub>2</sub>).

On heating, the water of hydration is given off between 30 and 97°C; this water may decompose the anhydrous salt. M.p. of KPbI<sub>3</sub> =  $349^{\circ}$ C, with I<sub>2</sub> beginning to separate at that point. The dihydrate is partially decomposed by pure water, forming PbI<sub>2</sub>; it is stable in aqueous solution only in the presence of a large excess of KI; quite soluble in acetone, yielding a bright-yellow solution.

#### REFERENCES:

C. H. Herty. Amer. Chem. J. <u>14</u>, 107 (1892).
W. Biltz. Ber. dtsch. chem. Ges. <u>40</u>, 2182 (1907).

## Lead (II,IV) Oxide

Pb<sub>3</sub>O<sub>4</sub>

## (Crystalline)

In contrast to the finely crystalline commercial  $Pb_3O_4$ , which is usually obtained by high-temperature oxidation of PbO or PbCO<sub>3</sub> with air, macroscopic crystals of  $Pb_3O_4$  are prepared by precipitating a solution of potassium plumbite with a solution of potassium plumbate in a strongly alkaline medium:

 $\begin{array}{rrrr} 2 \, K_2 Pb(OH)_4 & + & K_2 Pb(OH)_6 = Pb_3 O_4 & + & 6 \, KOH & + & 4 \, H_2 O \\ & & 706.9 & 387.5 & 685.6 \end{array}$ 

A 0.1 M  $K_2Pb(OH)_6$  solution is prepared according to one of the two procedures given in the Hydroxy Salts section(Part III, Section 2) for the preparation of the compound. The alkali concentration is adjusted to about 9 N.

At the same time, twice that volume of 0.1 MK\_Pb(OH)\_ solution ( $\approx 9$  N in alkali) is prepared by precipitating lead hydroxide with KOH from the appropriate quantity of lead acetate solution. The precipitate is suction-dried on a fritted glass filter, washed until the wash water gives a neutral reaction, and dissolved in strong KOH. All these steps must be performed in the cold to avoid decomposition (formation of PbO). The K salts are preferred to the Na salts because of their greater solubility. After filtration, the K<sub>2</sub>Pb(OH)<sub>4</sub> and K<sub>2</sub>Pb(OH)<sub>6</sub> solutions are thoroughly mixed in a 2:1 ratio and allowed to stand at room temperature until crystallization occurs. In order to increase the number of crystallization nuclei some glass wool should be placed in the vessel prior to the run. After some time, Pb<sub>3</sub>O<sub>4</sub> settles out. The completion of settling requires several days. Part of the product collects on the bottom of the vessel as a fine, red crystalline powder; the other part settles on the walls and on the glass wool in the form of large, rod-shaped and shiny crystals. Following crystallization, most of the mother liquor is decanted and the precipitate is suction-dried on a fritted glass filter. The crystals are washed with absolute alcohol until the filtrate is no longer alkaline and are dried in a vacuum desiccator over KOH. The Pb3O4 is then analytically pure.

See Clark, Schieltz and Quirke for the preparation of still larger single crystals from  $PbO_2$  and NaOH in the presence of  $H_2O$ , using a steel bomb at 355 to 375°C.

SYNONYMS:

Plumbous plumbate; trade name "red lead."

PROPERTIES:

Chemical behavior similar to that of the ordinary, finely crystallized material.

d 9.07. Tetragonal crystals, probable space group D<sup>7</sup><sub>2h</sub>.

**REFERENCES:** 

- G. Grube. Z. Elektrochem. 28, 273 (1922).
- M. Straumanis. Z. phys. Chem. (B) 52, 127 (1942).
- G. L. Clark, N. C. Schieltz and T. T. Quirke. J. Amer. Chem. Soc. 59, 2305 (1937).

## Lead (IV) Oxide

PbO<sub>2</sub>

I.  $2 \operatorname{Pb}(\operatorname{CH}_3\operatorname{COO})_2 + \operatorname{Ca}(\operatorname{OCl})_2 + 4 \operatorname{NaOH} =$  $\begin{array}{c} (3 \operatorname{H}_2\operatorname{O}) \\ 758.7 \end{array}$   $143.0 \qquad 160.0 \\ 2 \operatorname{PbO}_2 + \operatorname{CaCl}_2 + 4 \operatorname{CH}_3\operatorname{COONa} + 2 \operatorname{H}_2\operatorname{O} \\ 478.4 \qquad 111.0 \qquad 328.2 \end{array}$ 

A solution of 20 g. of  $Pb(CH_3COO)_2 \cdot 3 H_2O$  in 50 ml. of distilled water is prepared and mixed with a solution of 10 g. of NaOH in 90 ml. of water.

Good agitation is required. At the same time, a hypochlorite solution is prepared from 14 g. of technical grade  $Ca(OCI)_2$  (effective Cl content 70-80%) or from double this amount of technical grade bleaching powder. Either compound is dissolved in 200 ml. of distilled water. After filtering, 80 ml. of this solution is added slowly and with stirring to the alkaline Pb salt solution. The mixture is then heated and boiled for a few minutes. As soon as the brown precipitate of PbO<sub>2</sub> settles out, a few milliliters of the supernatant liquid are tested with a few drops of hypochlorite solution for the completion of oxidation. If further PbO<sub>2</sub> precipitates, an additional 10 ml. of hypochlorite is observed.

The dark, fine crystals of PbO<sub>2</sub> are washed 5 or 6 times with water. Then the precipitate is stirred with 50 ml. of 3 N HNO<sub>3</sub> in order to remove any Ca or Pb salts or Pb(OH)<sub>2</sub> which might have been formed. After washing several times with hot water, the precipitate is transferred to a Büchner funnel, thoroughly washed again, suction-filtered, and dried in a vacuum desiccator over  $P_2O_5$ . Even after a long time in the desiccator the product still contains small amounts of water which can be removed completely only by heating for 1.5 hours in a stream of  $O_2$  at 160°C. The PbO<sub>2</sub> is then analytically pure; the yield is about 85%.

II. 
$$Pb(CH_{3}COO)_{4} + 2H_{2}O = PbO_{2} + 4CH_{3}COOH$$
  
443.4 36.0 239.2 240.2

To prepare a particularly active compound for special oxidative reactions (e.g., organic reactions), the following procedure is used: 50 g. of  $Pb(CH_3COO)_4$  is carefully broken up and ground in a centrifuge tube with 460 ml. of water until all of the lead tetraacetate is hydrolyzed to  $PbO_2$ . The suspension is then centrifuged for 10 minutes, and the sediment is stirred up again with 460 ml. of water and centrifuged. This process is repeated four times. The last supernatant should give a neutral reaction with litmus. Finally the  $PbO_2$  is stirred with 50 ml. of water, suction-dried and washed with an additional 50 ml. of water. When the precipitate on a fritted glass filter is just barely moist, it is washed slowly four times with acetone, using 25 ml. each time, and thereafter four more times with absolute ether, again using 25-ml. portions. The  $PbO_2$  acquires a bright, coffee-brown color at this point. It is immediately dried in a vacuum desiccator. Yield: 23 g. (92% of theoretical).

SYNONYM:

Lead dioxide, erroneously referred to as "lead superoxide" in the older literature.

PROPERTIES:

Formula weight 239.21. A dark brown, heavy, microcrystalline powder with strongly oxidizing properties. Decomposes on heating above  $344^{\circ}$ C, yielding O<sub>2</sub> and forming Pb<sub>3</sub>O<sub>4</sub> and PbO; d 8.9-9.2.

Insoluble in water; quite soluble in mineral acids, forming Pb (IV) salts; even more soluble in hot, concentrated alkalis, forming hexahydroxyplumbates.

Crystal structure, C 4 type.

REFERENCES:

- I. L. C. Newell and R. N. Maxson in H. S. Booth. Inorganic Syntheses, Vol. I, p. 45, New York-London, 1939; H. F. Walton. Inorganic Preparations, New York 1948, p. 141; J. Krustinsons. Z. Elektrochem <u>40</u>, 246 (1934).
- II. R. Kuhn and I. Hammer. Chem. Ber. 83, 413 (1950).

## Sodium Metaplumbate

## Na<sub>2</sub>PbO<sub>3</sub>

 $\begin{array}{rcl} Na_{2}Pb(OH)_{6} &=& Na_{2}PbO_{3} \ + \ 3 \ H_{2}O \\ 355.3 & 301.2 & 54.1 \end{array}$ 

Pure Na<sub>2</sub>Pb(OH)<sub>s</sub> (see section on Hydroxy Salts for method of preparation) is heated over an open flame at 300°C in either a

758

round-bottom flask or a test tube, using an aspirator. A very good vacuum must be provided. Water is evolved and yellow  $Na_2PbO_3$  is formed.

PROPERTIES:

Yellow powder; more stable in air than the hydroxy salt. Darkens on further heating and decomposes at  $700^{\circ}$ C with release of oxygen. Hydrolyzed by hot water, forming PbO<sub>2</sub>.

**REFERENCES:** 

G. Grube. Z. Electrochem. <u>28</u>, 273 (1922); see also A. Simon. Z. anorg. allg. Chem. <u>177</u>, 109 (1929).

#### Sodium Orthoplumbate

#### Na<sub>4</sub>PbO<sub>4</sub>

 $PbO_2 + 2Na_2O = Na_4PbO_4$ 239.2 124.0 363.2

Preparation from PbO<sub>2</sub> and Na<sub>2</sub>O is completely analogous to the method of producing Na<sub>4</sub>SnO<sub>4</sub> from SnO<sub>2</sub> and Na<sub>2</sub>O. The conditions described above (p. 739)—necessity of excluding CO<sub>2</sub> and water vapor, prior pulverization of the reaction mixture, reaction by heating under vacuum in a magnesia vessel—are exactly the same in the preparation of the Pb salt. A furnace temperature of 400°C suffices, however, since Pb compound forms more easily. The conversion is quantitative. At this temperature, the vaporization of Ag from the protective tube, which sometimes causes contamination of the Sn reaction, is precluded. Pure, anhydrous PbO<sub>2</sub> is required as the starting material.

PROPERTIES:

Bright yellow, crystalline solid; hygroscopic. Reacts vigorously with water, producing  $PbO_2$ , probably with  $Na_2Pb(OH)_6$  as an intermediate.

**REFERENCE:** 

E. Zintl and W. Morawietz. Z. anorg. allg. Chem. 236, 372(1938).

## Calcium Orthoplumbate

#### Ca<sub>2</sub>PbO<sub>4</sub>

 $2 \operatorname{CaCO_3} + \operatorname{PbO} + \frac{1}{2} \operatorname{O_2} = \operatorname{Ca_2PbO_4} + 2 \operatorname{CO_2} \\ 200.2 \qquad 223.2 \qquad 351.4$ 

Equal weights of CaCO<sub>3</sub> and PbO are mixed well and heated in a combustion tube to medium red heat (about 800°C), being careful to exclude  $CO_2$ . The temperature should not exceed 850°C, for otherwise the reaction will not be quantitative because of the high oxygen vapor pressure over the Ca<sub>2</sub>PbO<sub>4</sub>. Cooled samples of the reaction mixture are periodically tested with dilute HNO, for the presence of carbonate ions. The filtrate from this test is treated with H<sub>2</sub>S to detect Pb ion. Heating is continued as long as appreciable amounts of the starting materials remain; if necessary, the material should be broken up and remixed to form a homogeneous mixture. The reaction is complete when the carbonate test is negative and when treatment with  $H_2S$  produces at most a weak brownish tint in the filtrate. Completion of the reaction may be determined more reliably by repeated volumetric analysis of the reaction products. Pure Ca\_PbO4 is an orange-red, spongy solid which may be removed easily from the tube and pulverized. A well-stoppered flask is required to protect the product from CO, in the air.

#### PROPERTIES:

Orange-red, microcrystalline powder which gradually turns brown and decomposes with the release of  $PbO_2$  upon exposure to air containing  $CO_2$ ; strong oxidizer. Liberates increasing quantities of  $O_2$  at temperatures above  $850^{\circ}C$ ; for this reason, the salt was previously used to separate pure  $O_2$  from the air since it could easily be regenerated by heating at lower temperatures.d 5.71.

Insoluble in water; any CO<sub>2</sub> dissolved in the water causes slow decomposition with release of CaCO<sub>3</sub>.

**REFERENCES:** 

K. Wedemeyer. Arch. Pharm. 230, 263 (1892).
G. Kassner. Arch. Pharm. 228, 109 (1890); 232, 375 (1894).

#### Lead Sulfide

PbS

#### (Crystalline)

While only amorphous or partially crystallized PbS precipitates when acid solutions of lead salts are treated with  $H_aS$ , the treatment of hot sodium plumbite solution with thiourea results in the formation of crystals of PbS of nearly uniform size.

To a solution of 75 g. of Pb(CH<sub>3</sub>COO)<sub>2</sub> · 3 H<sub>2</sub>O in one liter of distilled water, just enough concentrated NaOH is added so that the Pb(OH), which forms is immediately redissolved as Na<sub>2</sub>Pb(OH). At the same time a solution of 17 g. of C. P. thiourea, dissolved in one liter of distilled water, is prepared. After filtering, equal volumes of the two solutions are mixed and heated in a beaker with continuous stirring until the liquid boils. The liquid turns brown between 38 and 40°C; at about 50°C, a mirror of PbS is deposited on the walls and bottom of the beaker. If further heating causes bumping, it may be expedient to transfer the liquid to another vessel. Complete precipitation of the PbS requires continued boiling for 10 minutes. The heavy crystalline precipitate is then filtered, washed free of alkali with cold water, and dried in an oven. The yield is quantitative. The pure compound gives a sharp x-ray pattern; microscopic examination shows a wellformed crystal habit.

For the preparation of crystalline PbS by heating amorphous PbS at 1800°C in a nitrogen stream, see Weigel.

PROPERTIES:

Formula weight 239.27. Lead gray, crystalline powder with a metallic glint; chemically identical to the ordinary amorphous or partially crystallized compound. M.p.  $1110^{\circ}$ C, d 7.48.

Crystal structure, B 1 type.

**REFERENCES:** 

J. Emerson-Reynolds. J. Chem. Soc. (London) 45, 162 (1884).

O. Weigel. Nachr. d. Gesellsch. d. Wiss. Göttingen, Math. Phys. Klasse <u>1906</u>, 8 Dec.; Z. phys. Chem. <u>58</u>, 293 (1907).

## Lead (IV) Sulfate

## Pb(SO<sub>4</sub>)<sub>2</sub>

Lead (IV) sulfate is produced at the anode upon electrolysis of approximately 80% sulfuric acid, using lead electrodes:

A large battery jar is used for the electrolysis. The cathode is suspended in a ceramic cup. The cathode consists of a coil of lead

tubing, the end of which are bent over the edges of the battery jar and are fitted with tubing for the passage of cooling water. (Rubber tubing should not be exposed to the ozone-rich oxygen produced at the anode because it will soon deteriorate and begin to leak.) Two rolled lead anodes are suspended in the jar at equal distances from the ceramic cathode cell. The jar and the cathode cup are then filled with concentrated sulfuric acid (d 1.7 to 1.8). The electrolysis proceeds at a current density of 2 to 6 amp./dm.<sup>2</sup>; the temperature at the anode is prevented from rising above 30°C by continual cooling of the acid at the cathode. If the current density is too low,  $PbSO_4$  is the primary product. Higher temperatures cause hydrolysis of the  $Pb(SO_4)_2$  formed, and yield  $PbO_2$ . Should brown flakes of PbO<sub>2</sub> appear at the anodes, they should be withdrawn, washed with an acidifed solution of NaNO<sub>2</sub>, and rubbed dry with sand. If the electrolysis is carried out correctly, the solution near the anodes becomes turbid, and soon a white scum of crude  $Pb(SO_A)_2$  deposits on the bottom of the jar. The pale green-yellow supernatant liquid is a solution of Pb(SO4); in sulfuric acid. It is advisable to start with a relatively large quantity of anode solution to compensate for evaporation occurring during the process; it should be remembered that the conductivity falls off in proportion to the amount of solution removed. With care, excessive resistance of the solution during the run may be reduced by dilution. To accomplish this, the solution is cooled as much as possible and then cold, dilute H<sub>2</sub>SO<sub>4</sub> is slowly poured down the side of the battery jar. The electrolysis is continued for several hours in order to achieve optimum yield (about 60% based on current). It is advisable to let the temperature at the anode rise to 40-50°C during the last 60 minutes, since this produces betterformed  $Pb(SO_4)_2$  crystals and increases the purity of the precipitate. To stop the reaction, the ceramic cup and the electrodes are removed from the jar, the salt deposit on the anode is scraped into the acid solution, and the nearly clear, pale green-yellow supernatant liquid is siphoned off into a flask fitted with a ground glass stopper. The  $Pb(SO_4)_2$  gradually precipitates on cooling and forms a granular crust. The precipitate is collected on a glass frit by suction filtration; any residual sulfuric acid is to a large extent removed by repeated pressing on clay and leaving it in a desiccator until an apparently dry salt is obtained. It is impossible to remove all the residual sulfuric acid. This salt does not change on prolonged exposure to dry air. The purity of the product ranges from 85-99%. Another fraction of 60-85% purity can be obtained by drying the anode mud; this fraction is still contaminated by PbSO 4

#### SYNONYMS:

Lead disulfate, plumbic sulfate.

PROPERTIES:

White to yellow-green crystalline powder; stable for long periods in dry air; indefinitely stable when stored away from light under concentrated  $H_2SO_4$ ; strong oxidant. Hydrolyzes to form PbO<sub>2</sub>; somewhat soluble in concentrated  $H_2SO_4$ , giving a pale green-yellow solution.

REFERENCE:

K. Elbs and F. Fischer. Z. Electrochem. 7, 343 (1900/01).

## Lead Azide

 $Pb(N_3)_2$ 

 $\begin{array}{rrrr} Pb(NO_{3})_{2} &+& 2\,NaN_{3} &=& Pb(N_{3})_{2} &+& 2\,NaNO_{3}\\ 331.2 & 130.0 & 291.2 & 170.0 \end{array}$ 

A solution of  $Pb(NO_3)_2$  is added to a solution of  $NaN_3$ , with continuous stirring. Vigorous stirring is necessary to prevent the formation of large crystals, since these may detonate upon later grinding. The crystalline precipitate is collected by suctionfiltration, washed with water and dried in a desiccator.

**PROPERTIES:** 

White, finely crystalline powder. Readily soluble in water. Detonates on shock or impact.

## Tetramethyliead

Pb(CH<sub>3</sub>)<sub>4</sub>

The easiest procedure, giving the best yields, is the Grignard reaction of PbCl<sub>2</sub> with a solution of CH<sub>3</sub>MgCl:

4 Mg + 4 CH<sub>3</sub>Cl (ether) 4 CH<sub>3</sub>MgCl 97.3 202.0

 $\frac{4 \text{ CH}_3 \text{MgCl}}{556.2} + \frac{2 \text{ PbCl}_2 \text{ (ether)} \text{Pb}(\text{CH}_3)_4}{267.4} + \frac{\text{Pb}}{207.2} + 4 \text{ MgCl}_2}$ 

A solution of  $CH_3MgCl$  in absolute ether is prepared in a manner analogous to that of  $CH_3MgBr$ . (See preparation of tetramethyltin,

p. 744, for apparatus and procedure.) Because of the low boiling point of  $CH_3Cl$  (-23.7°C), the gas should be introduced only gradually so that the ether remains just at the boiling point; otherwise, appreciable losses of  $CH_3Cl$  are unavoidable. About 4-5 hours are required to completely dissolve 1 gram-atom of Mg. The solution is then refluxed for 30 minutes on a water bath, in the same manner as for the  $CH_3MgBr$  solution.

The rest of the procedure must be carried out under an efficient hood because the resulting  $Pb(CH_3)_4$ , like all alkylleads, is extremely toxic. A one-liter, three-neck flask is fitted with a high-efficiency condenser, a sealed stirrer and a straight filling adapter, which may be closed by a ground glass stopper. (The same apparatus may also be used for the preparation of the CH<sub>3</sub>MgCl solution.) The condenser is connected to a CaCl, tube to eliminate atmospheric moisture. Ground glass and rubber connections must be carefully made since otherwise the yield will be considerably reduced as a result of the extreme volatility of the  $Pb(CH_3)_4$  in ether. Small portions of finely powdered PbCl<sub>2</sub> (139 g. total) are gradually added to the flask containing the ether solution of CH\_MgCl; agitation must be continuous and the flask must be cooled with water. The filling aperture is only momentarily unstoppered. The mixture is then refluxed for 4-5 hours until the reaction is complete; after cooling, distilled water is added to the liquid, drop by drop, until the layers separate. The ether layer is siphoned off and, after drying over CaCl, for several hours, distilled in an efficient fractionating column. The solvent is distilled off at atmospheric pressure on an oil bath. A small flask is then filled with the residue and carefully fractionated in a column, using an oil bath for heating. Under no circumstances should an open flame be used since local overheating of the  $Pb(CH_3)_4$  in contact with the hot glass may cause explosive decomposition. If the crude product contains appreciable amounts of trimethyllead, a dark lead mirror appears on the walls of the flask because of decomposition of that compound. The fraction coming over between 105 and  $115^{\circ}C$  is refractionated, yielding about 25 g. of pure Pb(CH<sub>3</sub>)<sub>4</sub>, b.p. 110°C (uncorr.) at 760 mm.

Tetramethyllead may be stored for years in a brown, glassstoppered bottle. Sealing in ampoules should not be attempted under any circumstances because of the danger of explosion (see above).

The compound is the starting material for production of free methyl radicals.

## PROPERTIES:

A clear, dense, strongly refractive liquid with a pleasantly sweet odor; extremely toxic; stable in air and water. M.p.  $-27.5^{\circ}$ C, b.p.  $110^{\circ}$ C. The vapor pressure at room temperature is exceptionally high so that, despite the high boiling point (over  $100^{\circ}$ C), rapid evaporation occurs, as in the case of benzene. Very volatile with ether. d ( $20^{\circ}$ C) 1.995.

Insoluble in water and 96% alcohol; miscible in all proportions with absolute alcohol, ether and other common organic solvents.

REFERENCE:

E. Krause and A. von Grosse. Die Chemie der metall-organischen Verbindungen [The Chemistry of Organometallic Compounds], Berlin, 1937, p. 389.

## Tetraethyllead

#### Pb(C<sub>2</sub>H<sub>5</sub>)<sub>4</sub>

The preparation is similar to that described for  $Pb(CH_3)_4$  and is carried out by reacting  $PbCl_2$  with an absolute ether solution of  $C_2H_5MgBr$ .

 $\begin{array}{rrrr} 4 \text{ Mg} &+ 4 \text{ C}_2 \text{H}_5 \text{Br} \quad \text{(ether)} \quad 4 \text{ C}_2 \text{H}_5 \text{MgBr} \\ 97.3 & 435.9 \end{array}$ 

 $\begin{array}{rrrr} 4 \ C_2 H_5 M g B r \ + \ 2 \ P b C l_2 & (ether) \\ 556.2 & 323.5 & 207.2 \end{array}$ 

A side reaction results in the formation of considerable quantities of triethyllead. This is best disposed of by brominating the mixture [including the  $Pb(C_2H_5)_4$ ] to form  $(C_2H_5)_3PbBr$ , which then reacts with the  $C_2H_5MgBr$  solution, yielding pure  $Pb(C_2H_5)_4$ . The  $C_2H_5MgBr$  solution is prepared in a manner similar to

The  $C_2H_5MgBr$  solution is prepared in a manner similar to that for  $CH_3MgBr$  solution (see p. 744). The principal difference is that, instead of gaseous  $CH_3Br$ , 136 g. of liquid  $C_2H_5Br$  is added to 1 gram-atom of Mg by means of a dropping funnel. The mixture is then refluxed on a water bath for 30 minutes.

The procedure for the reaction of the Grignard solution with  $PbCl_2$  is carried out as in the preparation of  $Pb(CH_3)_4$  (see preceding preparation). The formation of undesirable triethyllead can be suppressed by adding the  $PbCl_2$  at room temperature, by allowing the reaction mixture to stand for a longer time after the completion of the reaction, and by refluxing for several hours.

After the distilled water has been added and the ether layer separated, it is desirable to remove the  $Pb(CH_3)_4$  by treating the crude ether solution of  $Pb(C_2H_3)_4$  at  $-70^{\circ}C$  with an ether solution of bromine until a persistent red-brown color appears. After

filtering out mechanically occluded impurities and drying over  $CaCl_2$ , the solution is again reacted with an equal amount of  $C_2H_5MgBr$ . If this treatment is omitted, considerable decomposition occurs in the subsequent distillation step, resulting in the precipitation of free lead. The dry ether solution of Pb( $C_2H_5$ )<sub>4</sub> is treated further, removing the solvent by distillation, using a column. The residue is then distilled twice under aspirator vacuum at 83°C (13.5 mm.). The yield is about 50% (based on PbCl<sub>2</sub>).

Tetraethyllead is stable for long periods of time if stored in brown, glass-stoppered bottles. Direct exposure to sunlight results in gradual decomposition.

PROPERTIES:

Colorless, mobile liquid with a pleasant, sweet odor; highly refractive; stable in air and water; toxic.

B.p. (13 mm.) 82°C. Decomposes, releasing lead, on further heating at atmospheric pressure. d (20°C) 1.653.

Insoluble in water and 96% alcohol. Miscible in all proportions with absolute alcohol, ether and other common organic solvents.

**REFERENCES:** 

- E. Krause and A. von Grosse. Die Chemie der metall-organischen Verbindungen [The Chemistry of Organometallic Compounds], Berlin, 1937, p. 389;
- G. Gruettner and E. Krause. Ber. dtsch. chem. Ges. <u>49</u>, 1415 (1916); see also Y. Tanaka and T. Kuwata. Chem. Zent. <u>1928</u>, I, 2593; Rochow, Hurd and Lewis. The Chemistry of Organometallic Compounds, New York, John Wiley and Sons, 1957, pp. 190-197.

## Neutral and Basic Lead Carbonate

PbCO<sub>3</sub>, 2 PbCO<sub>3</sub> · Pb(OH)<sub>2</sub>

## (Crystalline)

Prepared by precipitation with urea from an aqueous solution of a Pb salt, using pressure and high temperature.

A mixture of 37.9 g. of  $Pb(CH_3COO)_2 \cdot 3H_2O$  and 11.2 g. of PbO is dissolved in 300 ml. of hot distilled water acidified with acetic acid. After cooling, the solution is treated with 6.0 g. of urea, filtered and heated for 12 hours at  $180^{\circ}C$  in a thick-wall reaction tube. The resulting mixture of  $PbCO_3$  and  $2PbCO_3 \cdot Pb(OH)_2$  is easily separated by elutriation. If the

starting solution is dilute, the formation of the basic compound is favored. Both products are visibly crystalline and show welldefined crystals under the microscope.

**PROPERTIES:** 

Formula weight of  $PbCO_3$ , 267.22; of  $2PbCO_3 \cdot Pb(OH)_2$ , 775.66. Chemical properties are the same as for the corresponding ordinary (amorphous or microcrystalline) compounds.

PbCO<sub>3</sub>: d (25°C) 6.524. Crystals are GeO<sub>2</sub> structure type. 2PbCO<sub>3</sub>  $\cdot$  Pb(OH)<sub>2</sub>: d (25°C) 6.694. Hexagonal crystals.

**REFERENCES:** 

A. Lemke and W. Biltz. Z. anorg. allg. Chem. <u>220</u>, 312 (1934). L. Bourgeois. Bull. Soc. min. 11, 221 (1888).

# Lead (IV) Acetate Pb(CH<sub>3</sub>COO)<sub>4</sub>

$Pb_{3}O_{4}$ -	$+ 8 CH_3 COOH = 1$	Pb(CH <sub>3</sub> C0	$OO)_4 + 2 Pb(C)$	$H_3COO)_2 +$	4 H <sub>2</sub> O
685.6	480.4	443.4	3)	3 H₂O) 758.7	72.1
$4 \operatorname{H_{2O}} + 4 \operatorname{(CH_3CO)_2O} = 8 \operatorname{CH_3COOH}$					
	72.1	408.4	480.4		
	2 Pb(CH <sub>3</sub> COO) <sub>2</sub> -	$\vdash$ Cl <sub>2</sub> =	Pb(CH <sub>3</sub> COO) <sub>4</sub>	$+ PbCl_2$	
	(3 H <sub>2</sub> O) 758.7	70,9	443.4	278.1	

The reaction of  $Pb_3O_4$  with  $CH_3COOH$  is carried out in a oneliter, three-neck flask fitted with a sealed stirrer and a thermometer; the third opening may be closed (not too tightly). The required amount of pure  $Pb_3O_4$  is finely pulverized in advance, dried at 200°C and left in a desiccator over  $P_2O_5$  until needed. The flask is charged with 550 ml. of glacial acetic acid and 170 ml. of pure acetic anhydride; the mixture is heated to 40°C, and then 300 g. of Pb<sub>3</sub>O<sub>4</sub> is slowly added with vigorous stirring without further external heating. During this procedure, the loosely fitting stopper is removed, but only for brief periods. The reaction is exothermic and the rate at which the  $Pb_{a}O_{4}$  is introduced is regulated so that the temperature in the flask remains under 65°C. Otherwise, the freshly formed lead tetraacetate is partially reduced by the acetic anhydride. Running water may also be used to prevent overheating of the flask. After most of the Pb<sub>a</sub>O<sub>4</sub> has been added, the temperature gradually falls, and toward the end of the reaction it may be necessary to heat the flask (but never over  $65^{\circ}$ C).
The clear solution is left to cool protected from atmospheric moisture, and large quantities of colorless  $Pb(CH_3COO)_4$  precipitate out. The supernatant liquid is decanted, and the precipitate is poured into a large Büchner funnel. The funnel is covered with a cardboard square or a tile to minimize the effects of atmospheric moisture during the very slow suction-filtration. The filter cake is washed several times with glacial acetic acid and dried on a clay tile in a desiccator. The resulting product, usually tinted rose or brownish because of the presence of small amounts of PbO<sub>2</sub>, can be further purified by recrystallization from hot glacial acetic acid. Even after prolonged desiccation a small amount of glacial acetic acid will be retained by the salt. The yield is about 150 g.

Another, less pure quantity of the salt may be recovered from the mother liquor, which is treated in the original reaction flask with dry  $Cl_2$  at 80°C until no further PbCl<sub>2</sub> precipitates. Good stirring must be used. The precipitate is filtered hot and washed with glacial acetic acid, and the solution is left to crystallize. During cooling, about 100 g. of Pb(CH<sub>3</sub>COO)<sub>4</sub> crystallizes out. However, it is contaminated with PbCl<sub>2</sub>. Pure lead tetraacetate can be obtained by repeated further recrystallization from glacial acetic acid.

The procedure may be varied by omitting the acetic anhydride (Dimroth and Schweizer). In this case, however, the maximum temperature must be held below  $60^{\circ}$ C since the water formed in the reaction is not bound and may hydrolyze the Pb(CH<sub>3</sub>COO)<sub>4</sub> at higher temperatures.

Lead (IV) acetate may be stored only if absolutely dry and when kept in well-closed ground glass bottles. It is used as a selective oxidant in organic syntheses.

SYNONYMS:

Lead tetraacetate, plumbic acetate.

PROPERTIES:

Colorless prismatic crystals, very sensitive to moisture. In the presence of moisture, decomposes hydrolytically to form brown PbO<sub>2</sub>. M.p. 175-180°C (some decomposition); d (17°C) 2.23. Hydrolyzed by water, forming PbO<sub>2</sub> and acetic acid. Dissolves in hot acetic acid without decomposition; slightly soluble in dry CHCl<sub>3</sub>, CCl<sub>4</sub> and C<sub>8</sub>H<sub>8</sub>.

**REFERENCES:** 

H. F. Walton. Inorganic Preparations, New York 1948, p. 138. J. C. Bailar, Jr. in H. S. Booth, Inorg. Syntheses, Vol. I, New York-London 1939, p. 47; see also: O. Dimroth and R. Schweizer, Ber. dtsch. chem. Ges. 56, 1375 (1923).

## Lead Thiocyanate

Pb(SCN)<sub>2</sub>

 $\begin{array}{rrrr} Pb(NO_3)_2 &+ \ 2\ KSCN &= \ Pb(SCN)_2 &+ \ 2\ KNO_3 \\ 331.2 & 194.4 & 323.4 & 202.2 \end{array}$ 

A solution of 150 g. of C. P.  $Pb(NO_3)_2$  in 1.5 liters of distilled water is prepared and filtered; at room temperature a filtered solution of 90 g. of C.P. KSCN in 840 ml. of distilled water is added with stirring. If the compounds are not available in sufficient purity, they must first be purified by recrystallization, as the properties of  $Pb(SCN)_2$  are greatly affected by the presence of trace metals. After about half of the thiocyanate solution has been added, the solution becomes cloudy and a dense white precipitate of  $Pb(SCN)_2$  begins to separate. To complete the crystallization, the mixture is left to stand for some time in the refrigerator and then filtered cold through a Büchner funnel. The precipitate is washed with ice-cold water and dried in the dark on an unglazed clay dish over CaCl<sub>2</sub>. The yield is approximately 90 g. of analytically pure  $Pb(SCN)_2$ . Lead thiocyanate is used as a starting material for the synthesis of  $(SCN)_2$ .

**PROPERTIES:** 

White, needlelike crystals, light sensitive; Decomposes with discoloration when heated beyond  $190^{\circ}$ C. d 3.82. Insoluble in cold water. Monoclinic crystals.

**REFERENCES:** 

- Z. Karaoglanov and B. Sagortschev. Z. anorg. allg. Chem. 202, 62 (1931).
- W. H. Gardner and H. Weinberger in H. S. Booth. Inorg. Syntheses, Vol. I, New York-London 1939, p. 84.

# **SECTION 14**

Boron

H. J. BECHER

#### Boron

I. According to Moissan, very impure amorphous boron, containing about 80-90% B, is obtained by the reaction of  $B_2O_3$  with magnesium. According to Kroll the optimum yields are obtained as follows: A fireclay crucible, approximately 20 cm. high and 16 cm. in diameter, is painted with a paste of ignited MgO and sintered MgCl<sub>2</sub> and dried in a low-temperature oven. A mixture of 110 g. of  $B_2O_3$ , 115 g. of Mg shavings (the use of Mg powder frequently leads to explosive reactions) and 94 g. of powdered S is placed in the crucible. The reaction is started with an ignition pellet, after which it proceeds vigorously. After the mixture has cooled, it is extracted in water and then in dilute HCl for a week. The residue is treated several times by heating with HF and HCl, washed with water and dried in vacuum at 100°C. The yields are variable, with a maximum of 46%.

II. According to Kiessling, pure boron can be made by reducing BBr<sub>3</sub> with H<sub>2</sub> at 800°C. The reaction takes place in the apparatus shown in Fig. 238. The BBr<sub>3</sub> is prepared by the method of Meyer and Zappner from Br<sub>2</sub> and commercial boron (usually 70-80% pure) (cf. the method described on p. 782). Thus, 15 g. of B is pressed into pellets, and the quartz tube b is filled with them. The tube is heated to  $700^{\circ}$ C and dry Br<sub>2</sub> is added in drops from dropping funnel  $\alpha$ . The resulting BBr<sub>3</sub> will then collect in trap c, which is cooled with an ice-salt mixture. After about 30 minutes, 5-10 ml. of BBr<sub>3</sub> will have accumulated. The addition of Br<sub>2</sub> is stopped and excess  $Br_2$  from b and c is flushed out with  $H_2$ . The resulting BBr<sub>3</sub> should be colorless. The H<sub>2</sub> flow is then adjusted to 2-4 bubbles per second, the temperature of the quartz tube heater is raised to  $750-800^{\circ}$ C, and the BBr<sub>2</sub> in c allowed to evaporate in the H<sub>2</sub> stream at ambient temperature. As a result elemental B precipitates in d. Unreacted BBr<sub>3</sub> recondenses in e, which is cooled with ice-salt mixture. When no further  $BBr_3$  is left in c, traps e and c are interchanged and the decomposition continued in d. When the BBr<sub>3</sub> is all reacted, more material is prepared by allowing fresh  $Br_2$  to drop into quartz tube b. One charge of 15 g. of B will

be sufficient for preparation of 75 ml. of BBr<sub>3</sub>. Boron that has precipitated in d appears to catalyze further decomposition. Therefore, this tube should not be emptied too early. The resulting B is washed and dried with hot H<sub>2</sub>O. According to Kiessling the composition is: 98.9% B, 0.04% Al, 0.1% Si and traces of O, H and Mg.



Fig. 238. Preparation of high purity boron. a—dropping funnel for Br<sub>2</sub>, protected against atmospheric moisture; b—quartz tube for B pellets (10 mm. in diameter and 700 mm. long); c—first condensation trap for BBr<sub>3</sub>, capacity about 25 ml.;d—quartz tube for reduction of BBr<sub>3</sub>; e—second condensation trap for BBr<sub>3</sub>, capacity 25 ml.; o—tubular electrical heaters, about 600 mm. long; p—joints cemented with picein; q—mercury seals; their design is shown enlarged next to the principal figure; it illustrates the seal for the top of the condensation flask.

PROPERTIES:

Atomic weight 10.82. Gray-brown to yellow-brown powder. M.p.  $2300^{\circ}$ C; d 2.3. Ignites in air at  $700^{\circ}$ C. Reacts violently with concentrated HNO<sub>3</sub>.

**REFERENCES:** 

- I. H. Moissan. Compt. Rend. Hébd. Séances Acad. Sci. <u>114</u>, 392 (1892); W. Kroll. Z. anorg. allg. Chem. <u>102</u>, 1 (1918).
- II. R. Kiessling. Acta Chem. Scand. 2, 707 (1948).
- For other procedures, cf. A. W. Laubengayer, D. T. Hurd, A. E. Newkirk and J. L. Hoard. J. Amer. Chem. Soc. <u>65</u>, 1924 (1943).

## Aluminum Boride AlB<sub>2</sub>, AlB<sub>12</sub>

AlB<sub>2</sub>

 $\begin{array}{rrrr} Al + 2 B = AlB_2 \\ 26.97 & 21.64 & 48.61 \end{array}$ 

Finely powdered pure boron and aluminum are mixed in stoichiometric proportions and placed in a graphite tube closed with a graphite stopper. The stopper has a few fine grooves through which the inside of the tube can be degassed. The filled graphite tube is placed inside a quartz tube which has been well prerinsed with helium, and the quartz tube is evacuated. The tube is heated overnight at 800°C. The graphite tube will then contain a gray powder. The x-ray analysis indicates the presence of AlB<sub>2</sub>, along with some graphite and B<sub>4</sub>C impurities. These impurities are visible and may be mechanically removed.

**PROPERTIES:** 

Dark-gray, finely crystalline material. Fairly resistant to dilute acids.

REFERENCES:

E.F. Felton. J. Amer. Chem. Soc. 78, 5977 (1956).F. Lihl and P. Jenitschek. Z. Metallkunde 44, 414 (1953).

AlB<sub>12</sub>

A mixture of 50 g. of  $B_2O_3$ , 75 g. of S and 100 g. of Al (all the reagents must be dry) is reacted in a fireclay crucible. After cooling, the melt is removed from the crucible and pulverized, and water is added. After elutriation of the slag, the reduced particles are sorted out from the residue, separated as far as possible from the slag, and treated with concentrated HCl until a brilliant black crystalline residue remains. The latter is treated with 40% HF in a Pt crucible, washed with water and left in HCl until gas evolution ceases. It is then filtered, washed and dried.

SYNONYM:

Tetragonal boron.

**PROPERTIES:** 

Very hard, stable black crystals.

REFERENCES:

H. Biltz. Ber. dtsch. chem. Ges. 41, 2643 (1908).

H. Lihl and P. Jenitschek. Z. Metallkunde 44, 414 (1953).

For information on many other metal borides, cf. the section on Alloys and Intermetallic Compounds.

## Diborane

#### $B_2H_6$

6 LiH + 47.70	$- 8 BF_3 \cdot O(C_2H_5)_2 =$ 1135.52	$\begin{array}{r} B_2H_6 + 6\\ 27.69 \end{array}$	562.56 LiBF4 +	8 (C <sub>2</sub> H <sub>5</sub> ) <sub>2</sub> O 592.96
3 LiBH4	$+ 4 BF_3 \cdot O(C_2 H_5)_2 =$	$2 B_2 H_6 +$	3 LiBF4 -	$+ 4 (C_2 H_5)_2 O$
65.37	567.76	55,38	281.28	296.48

Diborane was first obtained from the mixture of boron hydrides resulting from the hydrolysis of magnesium boride; later it was produced by spark discharge in mixtures of BCl<sub>3</sub> or BBr<sub>3</sub> with  $H_2$  [1, 2, 3]. It now can be produced more easily and in larger quantities by the reaction of LiH, NaH or alkali borohydrides with BF<sub>3</sub> diethyl etherate [4]. To obtain good yields, the alkali hydrides must be very finely powdered. Since alkali hydrides are hygroscopic and difficult to grind, the use of alkali borohydrides, which are fine powders to start with, has certain advantages for laboratoryscale synthesis. On the other hand, LiH is a particularly economical starting material for the production of larger quantities of B<sub>2</sub>H<sub>8</sub>.

The procedure to follow can be applied regardless of whether an alkali hydride or alkali borohydride is used. The reaction vessel  $\alpha$  (Fig. 239), which can have a capacity of 0.5 to 2 liters, depending on need, is used. The cold finger b acts as a reflux condenser. From b, the product  $B_2H_6$  is passed through four successive cold traps  $f_1$  to  $f_4$ . Ground joints and stopcocks should be greased with silicone lubricant, but in quantitative work it is best to use mercury seals. The four traps are connected to a storage flask which in turn



Fig. 239. Preparation of  $B_2H_6$ . *a*-reaction vessel; *b*-cold finger;  $f_1$  to  $f_4$ -traps;  $h_1$  to  $h_9$ -stopcocks;  $v_1$  to  $v_4$ -mercury-filled pressure release valves.

is connected to a high-vacuum pump. All the equipment must be thoroughly dried prior to the run. The finely divided hydride or borohydride is placed in the reaction vessel and suspended in some anhydrous ether, and the required quantity of BF<sub>3</sub> etherate is placed in the dropping funnel. Use double the stoichiometric quantity of BF<sub>3</sub> etherate. The entire apparatus is thoroughly flushed with dry, purified N<sub>2</sub>; the cold finger b and trap  $f_1$  are cooled to  $-78^{\circ}$ C and traps  $f_2$  and  $f_3$  to the temperature of liquid N<sub>2</sub>. When all the equipment is properly flushed with N<sub>2</sub>, stopcocks  $h_1, h_5, h_7, h_8$  and  $h_8$  are closed and the BF<sub>3</sub> etherate is added slowly in drops to the hydride, using constant, vigorous stirring. At the same time the reaction vessel is heated to  $60^{\circ}$ C. When gas generation ends the B<sub>2</sub>H<sub>e</sub> is flushed into the traps (using  $N_2$ ), where it freezes out at  $-196^{\circ}C$ . After the reaction, some ether and a trace of  $B_2H_6$  will be found in  $f_1$ . The product will be largely in  $f_2$ , with a smaller amount in  $f_3$ . Now stopcocks  $h_2$  and  $h_3$  are closed and the cooling bath under trap  $f_1$  is replaced by a Dewar flask filled with liquid N<sub>2</sub>. This trap is used to collect the residues forming when B<sub>2</sub>H<sub>6</sub> is purified by fractional condensation. To purify, traps  $f_2$  to  $f_4$  are evacuated and the  $B_2H_6$  condensed in  $f_2$ . Then trap  $f_3$  is cooled to  $-100^{\circ}C$  and trap  $f_4$  to -196°C, and by slow heating the contents of  $f_2$  are transferred into  $f_3$  and  $f_4$ . After a single fractionation,  $f_4$  will contain pure diborane. The fractionation can be repeated with  $f_3$  cooled to  $-140^{\circ}$ C.

To prevent ignition on disassembly of the apparatus, the flask is flushed again with N<sub>2</sub>, which can escape via  $v_2$  if stopcocks  $h_2$  and  $h_3$  are left open. Some methanol is added dropwise to the reaction flask and the contents of trap  $f_1$  are allowed to thaw. The N<sub>2</sub> will then contain enough methanol vapor to render harmless all B<sub>2</sub>H<sub>6</sub> residues in  $f_1$ .

#### PROPERTIES:

Spontaneously igniting gas with strong, characteristic odor. M.p.  $-165.7^{\circ}$ C, b.p.  $-92.5^{\circ}$ C. Reacts quickly with water to form  $H_{3}BO_{3}$  and  $H_{2}$ .

**REFERENCES:** 

- A. Stock, E. Wiberg and H. Martin. Z. anorg. allg. Chem. <u>188</u>, 32 (1930) and earlier reports of Stock et al.
- H. I. Schlesinger and A. B. Burg. J. Amer. Chem. Soc. <u>53</u>, 4321 (1931).
- 3. A. Stock and W. Sütterlin. Ber. dtsch. chem. Ges. 67, 407 (1934).
- 4. H. I. Schlesinger, H. C. Brown, J. R. Gilbreath and J. J. Katz. J. Amer. Chem. Soc. 75, 195 (1953).

#### Lithium Borohydride

#### LiBH<sub>4</sub>

#### I. $4 \operatorname{LiH} + \operatorname{BF}_3 \cdot \operatorname{O}(\operatorname{C}_2\operatorname{H}_5)_2 = \operatorname{LiBH}_4 + 3 \operatorname{LiF} + \operatorname{O}(\operatorname{C}_2\operatorname{H}_5)_2$ 31.8 141.4 21.8 77.8 74.1

A steel autoclave, provided with a threaded, removable head, is filled with finely powdered LiH, and sufficient absolute ether is added to cover the LiH with a thick layer. Then about 2/3 of the stoichiometric amount of BF<sub>3</sub> ether is added. The autoclave is closed at once, since the reaction starts immediately. The reactants are heated at 120-130°C for several hours. After the autoclave has cooled, it is opened, and its contents are diluted with liberal amounts of ether and transferred to a flask. The ether solution is then decanted and the residual solvent distilled off. The LiBH<sub>4</sub> product is recrystallized from absolute ether, taking care to exclude moisture. The LiBH<sub>4</sub> crystallizes with one mole of ether of crystallization; this can be removed in vacuum at 33°C.

II.

The apparatus shown in Fig. 240 is used. The required amount of  $B_2H_6$  is condensed in trap  $f_1$  at  $-196^{\circ}C$  under an  $N_2$  blanket. Then the trap is connected to the apparatus and the dry reaction vessel is filled with 10 g. of finely divided LiH and 400 ml. of absolute ether. Stopcock  $h_4$  is opened to allow  $N_2$  to enter. The latter can initially escape via  $v_2$  and later, after the stopcocks at trap  $f_1$  and  $h_2$  have been opened, via  $v_1$ . The apparatus is thoroughly flushed with  $N_2$ ; then  $h_4$  and  $h_3$  are closed. The Dewar flask  $f_1$  is removed from trap  $f_1$ , and trap  $f_2$  is then immersed in a Dewar flask at  $-196^{\circ}C$ . As a result,  $B_2H_6$  will slowly evaporate from  $f_1$  to  $f_2$ . Any



Fig. 240. Preparation of LiBH<sub>4</sub> from LiH and B<sub>2</sub>H<sub>6</sub>.  $J'_1, J'_2$ —traps;  $h_1$  to  $h_4$ —stopcocks;  $u_1.u_2$ —pressure release valves.

entrained N<sub>2</sub> can escape via  $v_1$ . The reaction vessel is well stirred while B<sub>2</sub>H<sub>6</sub> passes through it. Any unreacted B<sub>2</sub>H<sub>6</sub> will condense in  $f_2$ . When  $f_1$  is empty, the last traces of B<sub>2</sub>H<sub>6</sub> are flushed out from  $f_1$  into  $f_2$  (use N<sub>2</sub>). Close  $h_2$ , open  $h_3$  and, by placing the Dewar flask at  $f_1$  and removing it from  $f_2$ , allow B<sub>2</sub>H<sub>6</sub> to evaporate in the opposite direction. If the LiH is sufficiently reactive, two such passes of B<sub>2</sub>H<sub>6</sub> through the reaction vessel, i.e., once in each direction, will suffice. Nitrogen is allowed to enter via  $h_1$ ; this will flush the remainder of the B<sub>2</sub>H<sub>6</sub> into  $f_1$ , where it will freeze out. The trap is then closed and the apparatus may be disassembled. The reaction vessel is rinsed with ether, the combined ether phase is decanted off, and the LiBH<sub>4</sub> is isolated by evaporating the solvent.

PROPERTIES:

Loose white powder. Hydrolyzes in the presence of atmospheric moisture. In the presence of  $H_2O$ , gives off  $H_2$  in a violent reaction. Approximately 2.5 g. of LiBH<sub>4</sub> will dissolve in 100 ml. of ether at 19°C. Used as a reducing agent in the same manner as LiAlH<sub>4</sub>.

REFERENCES:

- H. I. Schlesinger. and H. C. Brown. J. Amer. Chem. Soc. <u>62</u>, 3429 (1940).
- G. Wittig and P. Hornberger. Z. Naturforsch. 6b, 225 (1951).
- H. I. Schlesinger, H. C. Brown, H. R. Hoekstra and L. R. Rapp. J. Amer. Chem. Soc. 75, 199 (1953).

## Sodium Borohydride

#### NaBH<sub>4</sub>

The reaction is carried out in a round-bottom, three-neck cylindrical flask. A mercury-seal Monel stirrer is placed in the central neck. The stirrer is equipped with five blades, arranged one over the other. The blade dimensions should be such that the stirrer can fit through the neck, but still fit the wall of the flask as closely as possible. A thermometer is placed in the second neck and a condenser on the third. The top of the condenser is equipped with a wye-tube adapter, one side of which connects to a dropping funnel and the other to a soda-lime drying tube. The flask is placed in an electric furnace, the top of which is covered with glass wool and an asbestos lid. The thermometer is removed, and the flask flushed with N<sub>2</sub> through this neck. Then 50 g. of NaH is rapidly added and 50 g. of B (OCH<sub>3</sub>)<sub>3</sub> is placed in the dropping funnel. The stirrer and the furnace are then turned on. As soon as the thermometer indicates

a temperature of 200°C in the flask, the ether is added dropwise at a uniform rate. The addition should require 20-40 minutes, during which the temperature is kept at 230-270°C. The stirring is continued at this temperature for another hour. The flask is then allowed to cool and thoroughly dried isopropylamine or liquid NH<sub>3</sub> is used to extract the NaBH<sub>4</sub> from the now solid mixture, which, however, should have become well pulverized as a result of the constant stirring. The extraction with isopropylamine is carried out by refluxing for a few minutes; the extraction with NH<sub>3</sub> is done by stirring the reaction product for a few minutes. In either case the extract is filtered through a fritted glass filter and the solvent is evaporated. The NaBH<sub>4</sub> remains as a fine white powder, with a purity of 90-96%. The yield is 86-94%. The NaBH<sub>4</sub> is purified by recrystallization from either isopropylamine or water (it forms a dihydrate).

For unknown reasons the nature of the NaH used exerts a great influence on the yield and purity of the final product. Therefore, it is best to check the suitability of the starting material by making a small-scale preparation first. If the product is unsatisfactory, the NaH is preheated to 250°C and a small quantity of impure NaBH<sub>4</sub> from a preceding run is added to start the reaction. The yield can also be improved by a more uniform rate of addition of the B (OCH<sub>3</sub>)<sub>3</sub>.

**PROPERTIES:** 

Fine, white crystals (cubic system). Decomposes slightly in neutral aqueous solutions, from which it can be partially recrystallized as a dihydrate. Rapidly hydrolyzes in acid solution. Stable up to  $400^{\circ}$ C.

**REFERENCES:** 

H. I. Schlesinger, H. C. Brown and A. E. Finholt. J. Amer. Chem. Soc. 77, 205 (1953).

## Sodium Trimethoxyborohydride

### NaHB(OCH<sub>3</sub>)<sub>3</sub>

 $\begin{array}{rcl} NaH &+& B(OCH_3)_3 &=& NaHB(OCH_3)_3 \\ 24.01 & 103.92 & 127.93 \end{array}$ 

A one-liter, round-bottom flask equipped with a reflux condenser is well dried and flushed with N<sub>2</sub>. Finely powdered NaH (43 g.) is added, followed by 230 g. of B (OCH<sub>3</sub>)<sub>3</sub>, slowly added from a dropping funnel on top of the condenser. The reaction, which begins at once, liberates a considerable amount of heat. After the addition of the ester, the contents are refluxed at 70 °C for several hours. This causes a fivefold increase in the volume of the product, which simultaneously becomes pure white. When the volume no longer increases, the reflux condenser is replaced by a downward condenser and the excess  $B(OCH_3)_3$  is distilled off. The yield is nearly quantitative. It is best, however, to pretest the available NaH in a small experimental run and, if necessary, modify the reaction

PROPERTIES:

Loose white powder. Stable in dry air; hydrolyzes slowly in moist air. Decomposes when heated to  $230^{\circ}$ C. Rapidly reacts with  $B_{2}H_{5}$  to form NaBH<sub>4</sub> and B(OCH<sub>3</sub>)<sub>3</sub>. Decomposed by alcohol, forming H<sub>2</sub>.

REFERENCE:

H. C. Brown, H. I. Schlesinger, I. Sheft and D. M. Ritter. J. Amer. Chem. Soc. 77, 192 (1953).

## **Borine Trimethylaminate**

## $BH_3 \cdot N(CH_3)_3$

LiBH <sub>4</sub> +	$- [N(CH_3)_3H]Cl =$	$BH_3 \cdot N(CH_3)_3$	+	LiCl	+	$H_2$
21.79	95.58	72.95		42.40		2.02

A 100-ml. three-neck flask, equipped with a stirrer, a reflux condenser and a dropping funnel, is used and 1.68 g. of  $[N(CH_3)_3 H]Cl$  is added to it. A solution consisting of 0.42 g. of LiBH<sub>4</sub> in diethyl ether is slowly introduced from the dropping funnel. If vigorously stirred, the reaction proceeds at room temperature. When the generation of H<sub>2</sub> diminishes, the contents are refluxed for another hour. All solvent is then distilled and the solid residue is transferred to a vacuum sublimation apparatus, where the BH<sub>3</sub>  $\cdot$  N (CH<sub>3</sub>)<sub>3</sub> is sublimed in vacuum at 40°C and collected in a cooled receiver. The yield is 85%.

PROPERTIES:

White hexagonal crystals. Stable. M.p. 94°C.

REFERENCES:

- G. W. Schaeffer and E. R. Anderson. J. Amer. Chem. Soc. <u>71</u>, 2143 (1949).
- A. B. Burg and H. I. Schlesinger. J. Amer. Chem. Soc. <u>59</u>, 780 (1937).

778

## Borazole

## (HBNH)<sub>3</sub>

Borazole was discovered when a mixture consisting of  $B_2H_6$  and  $NH_3$  (or the addition product  $B_2H_6 \cdot NH_3$ ) was heated in a sealed tube [1, 2]. It can also be produced by pyrolysis of a mixture of LiBH<sub>4</sub> and NH<sub>4</sub>Cl [3]. The best yield is obtained by reducing (ClBNH)<sub>3</sub> (see below) with LiBH<sub>4</sub> in n-butyl ether. This reaction also yields  $B_2H_6$ . For details see the literature [4].

SYNONYM:

s-Triazaborane.

**REFERENCES:** 

- 1. A. Stock and E. Pohland. Ber. dtsch. chem. Ges. <u>59</u>, 2215 (1926).
- 2. E. Wiberg and A. Bolz. Ber. dtsch. chem. Ges. 73, 209 (1940).
- 3. G. W. Schaeffer, R. Schaeffer and H. I. Schlesinger. J. Amer. Chem. Soc. 73, 1612 (1951).
- R. Schaeffer, M. Steindler, L. Hohnstedt, H. S. Smith, Jr., L. B. Eddy and H. I. Schlesinger, J. Amer. Chem. Soc. <u>76</u>, 3303 (1954).

## s-Trichloroborazole

### (ClBNH)<sub>3</sub>

 $3 \text{ NH}_4\text{Cl} + 3 \text{ BCl}_3 = (\text{ClBNH})_3 + 9 \text{ HCl}$ 160.49 351.57 183.88 328.19

A three-neck, two-liter flask equipped with a mercury-seal stirrer and a reflux condenser is used. A cold finger is inserted on top of the condenser and is cooled with a mixture of Dry Ice and acetone. A similar finger, cooled to -78 °C, is inserted in the third neck. During the reaction, BCl<sub>3</sub> is introduced through this neck; the reagent condenses on the cold finger and thus is added to the flask dropwise. The flask is filled with a mixture of 50 g. of dry NH<sub>4</sub>Cl and powdered glass with 400 ml. of chlorobenzene. The flask is then heated to 140-150 °C, the inlet cold finger is connected to a trap with BCl<sub>3</sub>, and the latter is allowed to evaporate at such a rate that adrop of liquid BCl<sub>3</sub> enters the reaction mixture every three seconds. After about five hours the rate of the initially very vigorous generation of HCl drops off. The addition of BCl<sub>3</sub> is

interrupted and the excess BCl<sub>3</sub> still remaining in the flask is allowed to recondense on the inlet cold finger for another hour. The cooling is then stopped and the excess BCl<sub>3</sub> distilled off. The liquid phase is siphoned out from the flask and centrifuged for further clarification, if necessary. The chlorobenzene is distilled until solid (ClBNH)<sub>3</sub> remains as a residue. The latter is purified by vacuum sublimation at 50-60°C. The yield is approximately 40%. If the recovered chlorobenzene is recycled back to the flask which still contains the solid residue and if fresh NH<sub>4</sub>Cl is added, the yield can be increased appreciably by further reaction.

PROPERTIES:

Colorless crystals, exceedingly sensitive to moisture. M.p.  $84^{\circ}C$ ; d (25°C) 1.58. Soluble in benzene, CCl<sub>4</sub> and other organic solvents.

#### REFERENCES:

- Ch. A. Brown and A. W. Laubengayer. J. Amer. Chem. Soc. <u>77</u>, 3699 (1955).
- R. Schaeffer, M. Steindler, L. Hohnstedt, H. S. Smith, Jr., L. B. Eddy and H. I. Schlesinger, J. Amer. Chem. Soc. <u>76</u>, 3303 (1954).

For preparation and properties of some other borazole derivatives, see E. Wiberg, Naturwiss. <u>35</u>, 182, 212 (1948); H. J. Becher and S. Frick, Z. anorg. allg. Chem. <u>295</u>, 83 (1958).

## Boron Trichloride

#### BCl<sub>3</sub>

Fairly large quantities of  $BCl_3$  can be produced by heating dilute borax with charcoal in a stream of  $Cl_2$  at temperatures of 400 to 700°C. In the laboratory it is more convenient to produce it from  $BF_3$  and  $AlCl_3$ .

I.

The reaction is carried out in the apparatus shown in Fig. 241. The lower flask has a capacity of one liter; the upper bulb, half that. Anhydrous  $AlCl_3(67 \text{ g.} = 0.5 \text{ mole})$  is placed in the lower flask. The inlet tube is connected to a BF<sub>3</sub> generator and the BF<sub>3</sub> flow adjusted in such a manner that 132 g. or two moles of BF<sub>3</sub> are added to the vessel over a period of 30 minutes. At the same time,



Fig. 241. Preparation of boron trichloride.

the lower flask is heated with an open flame. Later the bulb is also heated. The BCl<sub>3</sub> distills off while the AlF<sub>3</sub> peels off from the walls of the flask as a light powder. The BCl<sub>3</sub> is cooled in a U tube cooled to  $-80^{\circ}$ C. Moisture is excluded by means of a drying tube. The impure product is shaken with some Hg and recondensed. The yield is 47 g.

II.

$$KBF_4 + AlCl_3 = BCl_3 + KF + AlF_3$$
  
125.9 133.4 117.2

An intimate mixture of 133.3 g. of  $AlCl_3$  and 62 g. of  $KBF_4$  is placed in the apparatus described above, which is then slowly heated in an oil bath to 150-170°C. Using the method described above, the  $BCl_3$  that has been distilled off is trapped and purified. The yield is poorer than that produced by method I.

PROPERTIES:

M.p.  $-107^{\circ}$ C, b.p. 12.5°C; d (0°C) 1.434. Colorless liquid, fuming in moist air.

**REFERENCES:** 

I and II: E. L. Gamble in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-London, 1950, p. 27. Synthesis of BCl<sub>3</sub> from amorphous B and Cl<sub>2</sub>: L. Gattermann. Ber. dtsch. chem. Ges. <u>22</u>, 195 (1889).

### Boron Tribromide

**BBr**<sub>3</sub>

I.

$$AlBr_3 + BF_3 = BBr_3 + AlF_3$$
  
266.7 67.8 250.6 84.0

One half mole (133.4 g.) of  $AlBr_3$  is distilled into the flask described for the preparation of  $BCl_3$ ;  $BF_3$  is added while heating

the flask. After some time the flask contents solidify. The heating and admission of  $BF_3$  are continued. As a result, the  $BBr_3$ product distills over into the  $-78^{\circ}C$  trap. The distillate still contains some  $Br_2$  which can then be removed by shaking with Hg. The  $BBr_3$  must be distilled for further purification. The yield is 87.7 g. (70%).

An alternative is to heat a mixture of  $AlBr_3$  and  $KBF_4$  in the flask itself, but this results in a much lower yield of  $BBr_3$ .

II.  $\begin{array}{ccc} B + 1^{1/2} Br_2 = BBr_3 \\ 10.8 & 239.8 & 250.6 \end{array}$ 

This procedure has been previously described in connection with the Kiessling method for preparing pure boron. If BBr<sub>3</sub> is desired, the apparatus shown in Fig. 238 is closed off by means of a drying tube inserted behind condensation trap c. Before the reaction the boron is thoroughly dried by prolonged heating at 600°C in a quartz tube flushed with a stream of H<sub>2</sub>. Then dropwise Br<sub>2</sub> addition is started, and the temperature of the reaction tube is raised to 700-750°C. The product BBr<sub>3</sub> is purified as in method I.

### **PROPERTIES:**

M.p.  $-46^{\circ}$ C, b.p. 90.8°C; d (0°C) 2.65. Colorless, easily hydrolyzed liquid.

### **REFERENCES:**

- I. E. L. Gamble in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-London, 1950, p. 27.
- II. Fr. Meyer and R. Zappner. Ber. dtsch. chem. Ges. <u>54</u>, 551 (1921); H. Menzel. Unpublished.

## Boron Triiodide

#### BI3

 $3 \operatorname{LiBH}_4 + 8 \operatorname{I}_2 = 3 \operatorname{LiI} + 3 \operatorname{BI}_3 + 4 \operatorname{H}_2 + 4 \operatorname{HI}_{65,37}_{2030,56}_{2030,56}_{401,55}_{1174,65}_{1174,65}_{8,06}_{8,06}_{511,67}$ 

The apparatus shown in Fig. 242 is used and 170 g. of  $I_2$  is added to flask a, while 5.1 g. of LiBH<sub>4</sub> is placed in addition bulb b under an N<sub>2</sub> blanket. The entire assembly is thoroughly flushed with N<sub>2</sub> by evacuating it several times through v and introducing

dry, oxygen-free N<sub>2</sub> through h. Then flask a is slowly heated on a bath until its inside temperature is between 120 and 125°C. The bulb b is turned in the joint so as to add small quantities of LiBH<sub>4</sub> to the flask. The reaction with I<sub>2</sub> is very vigorous. Trap  $J_1$  is cooled to 0°C, trap  $J_2$  to -78°C, and trap  $J_3$  to -196°C. At the end of the reaction, the product in a is sublimed into trap  $J_1$ , which is cooled to -78°C. Working under N<sub>2</sub>, trap  $J_1$  is disconnected at dand g and closed with ground glass stoppers. Pure CS<sub>2</sub> is added



Fig. 242. Preparation of boron triiodide. a) reaction flask; b) addition bulb; c) thermometer; d, g, e, vground joints;  $f_1$  to  $f_3$ ) low-temperature traps; h) stopcock.

through e in order to dissolve the sublimate. Then Hg and Zn dust are added in order to reduce any entrapped I<sub>2</sub>. After it has become colorless, the solution is left to stand, and a small glass wool plug is placed in joint g, connected to a sublimation flask, which in turn is fused to a sublimation tube with three bulb enlargements. The CS<sub>2</sub> solution is carefully poured into the sublimation flask through the glass wool plug. The entire sublimation assembly is then evacuated (high vacuum), thus vaporizing the solvent. The sublimation flask is detached from the remainder of the apparatus and the impure BI<sub>3</sub> is slowly sublimed into the first bulb, then into the second, etc. The final product is pure white and crystalline. If the reaction is carried out in hexane in the presence of an excess of LiBH<sub>3</sub>, it will go at room temperature.

PROPERTIES:

Colorless shiny crystals when completely pure. M.p. 49.9°C. Soluble in CS<sub>2</sub>. Unstable in air.

**REFERENCES:** 

W. C. Schumb, E. L. Gamble and M. D. Banus. J. Amer. Chem. Soc. 71, 3228 (1949). E. G. Hofling, Thesis, Stuttgart, 1956.
T. Renner, Angew. Chem. <u>69</u>, 478 (1957).

## Boron Trifluoride Dihydrate

BF<sub>3</sub> · 2 H<sub>2</sub>O

 $\begin{array}{r} BF_3 + 2 H_2 O = BF_3 \cdot 2 H_2 O \\ 67.8 \quad 36.0 \quad 103.8 \end{array}$ 

This compound is best prepared by adding the calculated amount of BF<sub>3</sub> to almost ice-cold water. The absorption is slow at first, but then proceeds more rapidly. The intermediate  $H_3BO_3$  separates out, but goes back in solution as BF<sub>3</sub> is added. A clear, mobile liquid, with the composition BF<sub>3</sub>·2H<sub>2</sub>O, is obtained. However, on fractional distillation under reduced pressure, partial decomposition occurs, with the formation of BF<sub>2</sub>OH · H<sub>2</sub>O and other products. If a small excess of one of the reagents is used in the preparation, pure BF<sub>3</sub>·2H<sub>2</sub>O can be separated out by partial freezing.

**PROPERTIES:** 

M.p. 5.9-6.1°C. d  $(20^{\circ}C)$  1.6315. Crystallizes in rhombic form, is isomorphous with NH<sub>4</sub>BF<sub>4</sub> and should probably be written as H<sub>3</sub>O [BF<sub>3</sub>OH]. The liquid does not attack glass and is stable at room temperature.

**REFERENCES:** 

- H. Meerwein. Ber. dtsch. chem. Ges. 66, 411 (1933).
- H. Meerwein and W. Pannwitz. J. prakt. Chem., N. S. <u>141</u>, 123 (1934).
- L. J. Klinkenberg and J. A. A. Ketelaar. Rec. Trav. Chim. Pays-Bas. <u>54</u>, 4, 959 (1935); J. S. McGrath. G. G. Stack and P. A. McCusker. J. Amer. Chem. Soc. <u>66</u>, 1263 (1944).

## Dihydroxyfluoroboric Acid

#### $H[BF_2(OH)_2]$

I.

 $\begin{array}{rrrr} H_{3}BO_{3} + 3 H_{2}O + 2 BF_{3} = 3 H[BF_{2}(OH)_{2}] \\ 61.8 & 54.0 & 135.6 & 251.4 \end{array}$ 

Two moles of  $BF_3$  are stirred into a suspension of one mole of  $H_3BO_3$  in three moles of  $H_2O$ . This results in a clear liquid which

distills at  $85^{\circ}$ C (25 mm.) and which is pure dihydroxyfluoroboric acid.

II. 
$$4 H_{3}BO_{3} + 2 BF_{3} = 2 H[BF_{2}(OH)_{2}] + 4 HBO_{2} + 2 HF$$
  
247,2 135,6 167,6 175,2 40

An excess of  $BF_3$  is added to a Pyrex vessel containing some boric acid. The  $BF_3$  is absorbed in an exothermic reaction, with the material in the vessel finally liquefying. The resulting dihydroxyfluoroboric acid can then be distilled off under atmospheric pressure.

This acid is also formed when  $BF_3$  is produced from  $CaF_2$ ,  $B_2O_3$  and concentrated  $H_2SO_4$ , or from  $B_2O_3$  and HF, and can be obtained at the end of  $BF_3$  generation by fractional distillation of the residue in the generator.

#### PROPERTIES:

Colorless, syrupy liquid. B.p.  $159-160^{\circ}C$  (745 mm.) When the liquid is cooled to  $0^{\circ}C$ , the compound becomes very viscous, but no definite freezing point can be observed. Does not attack glass. Hydrolyzed by water either to HBO<sub>2</sub> or to H<sub>3</sub>BO<sub>3</sub>.

**REFERENCES:** 

- I. J. S. McGrath, G. G. Stack and P. A. McCusker. J. Amer. Chem. Soc. <u>66</u>, 1263 (1944).
- II. F. J. Sowa, J. W. Kroger and J. A. Nieuwland, J. Amer. Chem. Soc. 57, 454 (1935); 59, 965 (1937).

### Boron Trifluoride Ammoniate

BF<sub>3</sub> · NH<sub>3</sub>

$$\begin{array}{rcl} \mathbf{BF_3} + \mathbf{NH_3} = \mathbf{BF_3} \cdot \mathbf{NH_3} \\ \mathbf{67.8} & \mathbf{17.0} & \mathbf{84.8} \end{array}$$

I. A slow stream of  $NH_3$  gas is led into a three-neck, one-liter flask. When the flask has been flushed, an equivalent amount of  $BF_3$  is allowed to enter the flask through a second neck. An excess of  $BF_3$  can be detected by formation of a white cloud at the flask outlet. The  $BF_3NH$  deposits in the flask as a white powder. II. A portion of  $BF_3$  diethyl etherate is diluted with twice its quantity of ether. The solution is placed in a flask, and a uniform stream of  $NH_3$  is introduced. The reaction must be conducted in the absence of air and with vigorous stirring. The product settles as a fine precipitate. After the reaction is complete, the ether is decanted and the residual  $BF_3 \cdot NH_3$  washed several times with fresh ether.

#### PROPERTIES:

White, crystalline material. M.p.  $163^{\circ}$ C.d ( $25^{\circ}$ C) 1.864. Soluble in water. Hydrolyzes slowly in water at  $0^{\circ}$ C. Decomposes above  $125^{\circ}$ C.

REFERENCES:

- I. A. W. Laubengayer and G. F. Condike. J. Amer. Chem. Soc. 70, 2274 (1948).
- II. Ch. A. Kraus and E. H. Brown. J. Amer. Chem. Soc. <u>51</u>, 2690 (1929).

## **Boron Trifluoride Etherate**

### $BF_3 \cdot O(C_2H_5)_2$

 $BF_3 + O(C_2H_5)_2 = BF_3 \cdot O(C_2H_5)_2$ 67.8 74.1 141.9

Two moles (135.6 g.) of  $BF_3$  are generated as described on p. 219. It is then introduced into a dry one-liter flask containing 148 g. of diethyl ether. Alternately, the  $BF_3$  may be frozen at liquid nitrogen temperature and added as a solid. The flask is cooled in ice-salt mixture and access of atmospheric moisture is prevented by closing off the flask with a CaCl<sub>2</sub> tube. The stream of  $BF_3$  should not be too fast because some of it will not be taken up by the ether in that case. After the end of the reaction a condenser is connected to the flask and the  $BF_3 \cdot O(C_2H_E)_2$  distilled off at 125°C into a receiver, preferably at reduced pressure. The etherate distills readily at 38°C (6 mm.).

PROPERTIES:

Colorless liquid. M.p. -60.4°C, b.p. 125-126°C; d (25°C) 1.125. n<sub>D</sub> 1.348. Hydrolyzes quite readily. Used in organic reactions and for reactions with alkali hydrides.

REFERENCES:

- V. Gasselin. Ann. Chim Phys. [7] 3, 5 (1894).
- G. F. Hennion, H. D. Hinton and J. A. Nieuwland. J. Amer. Chem. Soc. 55, 2858 (1933);

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- H. Meerwein and W. Pannwitz, J. prakt. Chem., N. S. <u>141</u>, 123 (1934).
- E. Wiberg and W. Mathing. Ber. dtsch. chem. Ges. 70, 690 (1937).
- A. W. Laubengayer and G. R. Finlay. J. Amer. Chem. Soc. <u>65</u>, 884 (1943).
- H. C. Brown and R. M. Adams. J. Amer. Chem. Soc. <u>64</u>, 2557 (1942).

### Boron (III) Oxide

### $B_2O_3$

VITREOUS B<sub>2</sub>O<sub>3</sub>

 $\begin{array}{rcl} 2 \, \mathrm{H_{3}BO_{3}} \, = \, \mathrm{B_{2}O_{3}} \, + \, 3 \, \mathrm{H_{2}O} \\ 123.6 & 69.6 & 54.0 \end{array}$ 

Recrystallized  $H_3BO_3$  (several times from water) is placed in a Pt crucible and dried in vacuum over  $P_2O_5$ , slowly raising the temperature to 200°C. This results in white, only slightly sintered  $B_2O_3$  which is readily powdered.

**PROPERTIES:** 

Amorphous, M.p. 294°C; d 1.84. The glasslike pieces that are obtained by allowing molten  $B_2O_3$  to solidify are brittle, very hard and hygroscopic.

**REFERENCES:** 

E. Tiede and A. Ragohs, Ber. dtsch. chem. Ges. <u>55</u>, 594 (1922). E. Tiede and P. Wulff, Ber. dtsch. chem. Ges. <u>56</u>, 656 (1923).

CRYSTALLINE **B**<sub>2</sub>**O**<sub>3</sub>

Fused boric acid is heated in a loosely stoppered Pyrex flask to  $225-250^{\circ}$ C; after this treatment the melt retains 8-15% water. Crystallization requires a few days. If a small amount of crystalline  $B_2O_3$  seed is added to the melt, which contains some water, crystallization occurs very rapidly and the water separates out. Alternatively, fused boric acid is heated to  $175^{\circ}$ C, at which point crystalline HBO<sub>2</sub> (I) separates. The material is then transferred to a closed tube and the temperature is raised to about 236°C, i.e., just below the m.p. of HBO<sub>2</sub> (I). As a result the latter evolves water and transforms to crystalline  $B_2O_3$ . PROPERTIES:

M.p. 450°C; d 2.42-2.46. Hexagonal crystals.

**REFERENCES:** 

L. McCulloch. J. Amer. Chem. Soc. 59, 2650 (1937).

## Boron (III) Sulfide

#### $B_2S_3$

This compound is synthesized by the method of Sliwinski in the apparatus shown in Fig. 243. Before the reaction the apparatus is flushed with dry H<sub>2</sub> until free of explosive oxyhydrogen mixture. Amorphous boron is placed in a Vycor tube and heated to a dull red. A stream of dry,  $CO_2$ -free H<sub>2</sub>S is passed over the boron. Molten boron sulfide condenses close to the point where the heat has been applied (point a, Fig. 243). Upon cooling, the material becomes transparent. Further downstream, at b, porcelainlike sulfide forms, while crystalline B<sub>2</sub>S<sub>3</sub> forms at c. The sublimation zone should not be cooled or the crystals will not be well formed. A steady stream of H<sub>2</sub>S is maintained throughout the entire reaction and controlled to give a flow of approximately 3 liters/hour.



Fig. 243. Synthesis of boron (III) sulfide.

According to Hoffmann it is possible to substitute iron (III) boride for the elemental boron. The iron boride (finely divided powder) is placed in the tube, where it reacts with the  $H_2S$ . The reaction starts at 200°C; the optimum temperature lies between 300 and 400°C. The resulting  $B_2S_3$  sublimes into the end of the tube where it forms bundles of fine, hairlike crystals. Vitreous and amorphous residues can be converted to the crystalline form by cautious heating. The procedure is suitable for the production of relatively large amounts of  $B_2S_3$ .

#### PROPERTIES:

Extremely readily hydrolyzed; decomposes even in moist air. The products of hydrolysis are  $H_3BO_3$  and  $H_2S$ . The crystalline form consists of white, brilliant, needlelike crystals. Also occurs in an amorphous form. Becomes viscous on heating and begins to melt at  $310^{\circ}C$ .

#### **REFERENCES:**

P. Sabatier. Compt. Rend. Hébd. Séances Acad. Sci. <u>112</u>, 862 (1891).
H. Moissan. Compt. Rend. Hébd. Séances Acad. Sci. <u>115</u>, 205 (1892).
S. Sliwinski. Thesis, Dresden, 1944.
J. Hoffmann. Z. anorg. allg. Chem. 66, 362 (1910).

## **Boron Nitride**

#### BN

I.

$$\begin{array}{rcrcrcrc} B_2O_3 & + & NH_2CONH_2 & = & 2 BN & + & CO_2 & + & 2 H_2O \\ 69.6 & & 60.0 & & 49.7 \end{array}$$

One part by weight of fused, finely divided  $B_2O_3$  is intimately mixed with 1.5-2 parts by weight of urea and ignited in a covered procelain crucible until it becomes red hot. The resulting material is ground and washed with water to which a few drops of HCl have been added. It is then filtered; the resulting boron nitride is ovendried. The yield is 34%, as  $B_2O_3$  evaporates during ignition.

II. 
$$B_2O_3 + 2 NH_3 = 2 BN + 3 H_2O_{69.6} - 34.0 - 49.7$$

One part of finely divided  $H_3BO_3$  is triturated with two parts of pure  $Ca_3(PO_4)_2$  and the mixture is dehydrated in a Hessian crucible. This produces a porous mixture, which is then placed in a crucible covered with a lid through which a clay tube, touching the bottom of the crucible, is passed. The crucible is ignited in a gasheated furnace. Simultaneously,  $NH_3$  is bubbled slowly through the crucible via the clay tube. When the reaction is complete the cooled crucible contents are stirred with some water and transferred to a beaker; sufficient HCl to dissolve the phosphate is added and the mixture is brought to a boil. The BN residue is washed with acidified water (by decanting) until the wash water is free of both  $Ca^{2+}$  and  $PO_4^{3-}$  ions. The product is filtered, washed once again with water and dried, first on a clayplate and later in vacuum. The yield is 80-90%, based on  $B_2O_3$ . Boron nitride synthesized either via method I or method II still contains a few percent of  $B_2O_3$ .

Boron tribromide is added dropwise to an excess of liquid  $NH_3$ , while an  $H_2$  stream is passed through the flask. The  $NH_3$  is then evaporated; the moisture-sensitive white residue is transferred to a combustion tube and heated slowly to  $750^{\circ}C$  is a stream of dry  $NH_3$ . The yield of pure BN is theoretical.

IV.  $BCl_{8} + 4 NH_{3} = BN + 3 NH_{4}Cl_{117.2} 68.1 24.8 160.5$ 

A gaseous mixture of  $H_2$  and  $BCl_3$  is combined with an excess of  $NH_3$  in the front section of a quartz tube, heated to  $600^{\circ}C$ . In the middle section of this tube the mixture is then subjected to a temperature which rises from 500 to  $1000^{\circ}C$ . When decomposition is complete, heating in the stream of  $NH_3$  is continued at  $1000^{\circ}C$ for another hour. This procedure also yields pure BN.

#### PROPERTIES:

Light, white powder. Hydrolyzes slowly in boiling water. M.p. above  $2800^{\circ}C$ ; d 2.34.

#### **REFERENCES:**

- I. M. Darmstadt. Liebigs Ann. Chem. 151, 256 (1869).
- II. L. Moser and W. Eidmann. Ber. dtsch. Chem. Ges. <u>35</u>, 536 (1902).
- III. A. Stock and W. Holle. Ber. dtsch. chem. Ges. 41, 2095 (1908).
- IV. Fr. Meyer and R. Zappner. Ber. dtsch. chem. Ges. <u>54</u>, 560 (1921).

## Sodium Orthoborate

### Na<sub>3</sub>BO<sub>3</sub>

Sodium orthoborate is formed by the reaction of  $B_2O_3$  with  $Na_2CO_3$  at temperatures above 680°C.

Do not heat above 950°C.

**REFERENCE** :

Carrière, H. Guiter and F. Thubert. Bull. Soc. Chim. France [5] <u>16</u>, 796 (1949).

## **Metaboric Acid**

### HBO<sub>2</sub>

 $H_{3}BO_{3} = HBO_{2} + H_{2}O_{61.8} + 43.8 + 18.0$ 

Metaboric acid exists in three modifications:

Modification III occurs when  $H_3BO_3$  is heated to  $80-100^{\circ}C$  for several days, either in an oven or in a dry air stream, until the calculated amount of water has been removed.

PROPERTIES:

Loose, white powder; rapidly crystallizes from solution. Rhombic crystals. M.p.  $176^{\circ}$ C.

Modification II can be obtained from modification III by heating the latter in a sealed ampoule at  $130-140^{\circ}$ C. Under these conditions the loose powder transforms to well-formed monoclinic crystals. The transformation requires several days. M.p.  $201^{\circ}$ C.

Modification I can be obtained from modification II by further heating the latter at  $140^{\circ}$ C in a sealed glass ampoule. The progress of the transformation can be followed by observing the structural change.

**PROPERTIES:** 

Modification I is the most stable; crystallizes in the cubic system. Precipitates from water very slowly. M.p. 236°C.

**REFERENCE** :

F. C. Kracek, G. W. Morey and H. F. Merwin. Am. J. Sci. <u>A35</u>, 143 (1938).

#### Sodium Metaborate

NaBO<sub>2</sub>

The hydrates  $NaBO_4 \cdot 4H_2O$ ,  $NaBO_4 \cdot 2H_2O$  and  $NaBO_2 \cdot \frac{1}{2}H_2O$ , as well as the anhydrous  $NaBO_2$  are present in the system  $NaBO_2$ - $H_2O$ . The tetra- and dihydrates crystallize from aqueous solution; the hemihydrate can be obtained by cautious decomposition of the dihydrate.  $NaBO_2 \cdot 4 H_2O$ 

Synthesis of this compound requires completely carbonate-free 50% sodium hydroxide. The hydroxide is prepared in accordance with the instructions in Part II, Section 2, Hydroxy Salts.

Carbonate-free 50% sodium hydroxide (50 g.) is placed in a ground glass flask with 69.6 g. of  $H_3BO_3$  and diluted with 85 ml. of  $H_2O$ . When crystallization is complete, the slurry is filtered and the crystals are rinsed on the filter with a very small quantity of ice-cold water and dried on a clay plate. The well-formed crystals still contain small inclusions of the mother liquor.

**PROPERTIES:** 

Triclinic crystals. d 1.743. Slightly soluble in water ( $0^{\circ}$ C). The solution hydrolyzes very readily.

 $NaBO_2 \cdot 2 H_2O$ 

 $\begin{array}{rrrr} Na_{2}B_{4}O_{7} \cdot 10 \ H_{2}O \ + \ 2 \ NaOH \ = \ 4 \ NaBO_{2} \cdot 2 \ H_{2}O \ + \ 3 \ H_{2}O \\ 381.4 \ & 80.0 \ & 407.4 \ & 54.0 \end{array}$ 

Carbonate-free 50% sodium hydroxide solution (cf. above) is added to  $Na_2B_4O_7 \cdot 10 H_2O$  and 4 parts  $H_2O$  in a ratio of 42:21:8 by weight. The mixture is heated on a steam bath until the material dissolves, and the flask is then slowly cooled. At temperatures above 54°C the dihydrate crystallizes out in the form of needles and flat prisms. While they are still hot, the needles are dried as well as possible by suction filtering through fritted glass. Do not rinse with water, as this leads to the formation of the tetrahydrate. The crystals are first dried on clay plates and then in a vacuum desiccator over KOH.

The dihydrate can also be prepared from the tetrahydrate by isothermal decomposition, using suitable drying agents.

PROPERTIES:

Triclinic crystals. d (25°C) 1.905.

**REFERENCES:** 

H. Menzel and H. Schulz. Z. anorg. allg. Chem. <u>251</u>, 167 (1943).
 W. C. Blasdale and C. M. Slansky. J. Amer. Chem. Soc. <u>61</u>, 117 (1936).

ANHYDROUS NaBO<sub>2</sub>

This can be prepared by stepwise degradation of the tetrahydrate in a vacuum desiccator to the hemihydrate. The latter is then completely dehydrated by heating to 200°C in vacuum over  $P_2O_5$ . Another procedure consists in slowly heating an equimolar mixture of Na<sub>2</sub>CO<sub>3</sub> and H<sub>3</sub>BO<sub>3</sub> in a Pt dish with a final melting step. On cooling, the clear melt solidifies to a crystalline mass.

**PROPERTIES:** 

Hexagonal crystals. M.p. 965°C; d 2.34. Water soluble.

REFERENCES:

H. Menzel and H. Schulz. Z. anorg. allg. Chem. 251, 167 (1943).

S. S. Coole, S. R. Scholes and C. R. Amberg. J. Amer. Ceram. Soc. <u>18</u>, 58 (1935).

## Sodium Tetraborate

 $Na_2B_4O_7$ 

### ANHYDROUS NaBO2

To prepare very pure, formula-weight borax as a primary standard, good quality commercial product is recrystallized three times from water and dried to constant weight in a vacuum desiccator over a desiccant with suitable vapor pressure. Recent studies have shown that the best desiccant consists of solid NaCl, sucrose and a saturated sucrose solution. The correct water content (47.21%) is achieved by drying a sample in a Pt crucible, first on a steam bath, then at 200°C, and finally between 700-800°C. To check for any further impurities, cf. I. M. Kolthoff, Gravimetric Analysis, II (pp. 97-98).

PROPERTIES:

Equivalent weight 190.27. Large, colorless, transparent crystals, which effloresce superficially when stored in dry air. Water is split off on heating; above  $350^{\circ}$ C, Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub> is formed.

**REFERENCES:** 

H. Menzel. Z. anorg. allg. Chem. 224, 10 (1935).

 $Na_2B_4O_7 \cdot 5H_2O$ 

A borax solution, saturated at about  $90^{\circ}$ C, is concentrated by heating on a constant temperature bath at  $65-70^{\circ}$ C while a stream

of dry air passes over the material. The procedure takes several days; well-formed crystals of the pentahydrate precipitate out. These are filtered at  $65^{\circ}$ C through a heated suction filter and dried as well as possible between filter papers.

**PROPERTIES:** 

Trigonal crystals. Also called jeweler's borax or, incorrectly, octahedral borax. At temperatures above 88°C at 10 mm. it converts to the dihydrate. d 1.81.

References:

H. Menzel. Z. anorg. allg. Chem. 224, 14 (1935).

Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub> · 4 H<sub>2</sub>O (SYNTHETIC KERNITE)

A mixture consisting of equal weights of the pentahydrate and the decahydrate is heated in a tightly closed container (either a sealed ampoule or a stoppered flask) at  $120^{\circ}$ C for 24 hours. The synthetic kernite crystallizes from the partly molten system as well-formed crystals. To separate the latter, the sealed tube is turned over (in the bath), thereby draining the mother liquor from the crystals. The container is opened and the crystals scraped out and dried between filter papers.

The transformation also takes place at 100°C, but it then requires several days.

If the tube is agitated in the bath from the very beginning of the run, the resulting crystals are much smaller. These must then be separated on a suction filter at  $100^{\circ}$ C. It is not advisable to rinse the crystals with hot water.

PROPERTIES:

Monoclinic prismatic crystals; ready cleavage along the crystal grain. d 1.908.

REFERENCES:

H. Menzel and H. Schulz. Z. anorg. allg. Chem. 245, 157 (1941).

ANHYDROUS Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub>

Anhydrous Na<sub>2</sub>  $B_4O_7$  can be obtained by heating the hydrated salt to above 400°C. The mass swells during heating. The salt, which at first is amorphous, can be crystallized by prolonged heating at 600-650°C.

#### **PROPERTIES:**

Formula weight 201.27. d 2.36. On rapid cooling, the anhydrous tetraborate solidifes from a melt in the form of a hard glass. When the cooling is slow, three distinct modifications can be detected by x-ray analysis. Their occurrence depends on the conditions employed.

REFERENCE:

I. M. Koltholff. J. Amer. Chem. Soc. <u>48</u>, 1447 (1926). H. Menzel. Z. anorg. allg. Chem. <u>224</u>, 19 (1935).

## **Sodium Pentaborate**

 $NaB_5O_8 \cdot 5H_2O$ 

 $\begin{array}{rl} Na_{2}B_{4}O_{7}\cdot 10\,H_{2}O\,+\,H_{3}BO_{3}\,=\,NaB_{5}O_{8}\cdot 5\,H_{2}O\,+\,NaOH\,+\,6\,H_{2}O\\ 381.4 & 61.8 & 205.1 \end{array}$ 

A mixture of 34 g. of borax and 34 g. of  $H_3BO_3$  is dissolved in 140 g. of water at 60-70°C, cooled with stirring to 25°C, and allowed to crystallize at that temperature. After decanting, the crystals are suction-dried, covered briefly with ice-cold water, and filtered again. The residue is left to stand for five hours with twice the quantity of absolute alcohol, filtered and rinsed with alcohol. The product is dried on clay plates.

**PROPERTIES:** 

Crystallizes in the form of bevelled, blunt prisms. Solubility  $(0^{\circ}C)$  9.24 g./100 g. solution.

REFERENCES:

- W. C. Blasdale and C. M. Slansky. J. Amer. Chem. Soc. <u>61</u>, 917 (1939).
- H. Menzel. Z. anorg. allg. Chem. 164, 52 (1927).
- A. Rosenheim and F. Leyser. Z. anorg. allg. Chem. 119, 24 (1921).

### Sodium Perborate

### $NaBO_3 \cdot 4 H_2O$

According to Tanatar, pure sodium perborate can be obtained by allowing  $H_2O_2$  to act on an alkaline solution of borax. Thus, 38.14 g. of borax and 8 g. of NaOH are dissolved in 265 g. of  $H_2O$ , and 45 g. of

Perhydrol is added. After some time, crystals of sodium perborate separate out. They are filtered and washed, first with cold  $H_2O$ , then with alcohol and ether.

PROPERTIES:

Formula weight 153.88. Crystallizes in large, transparent monoclinic prisms. Melts with decomposition. Solubility  $(20^{\circ}C)$  1.17 g./100 g. H<sub>2</sub>O. This compound can also be written as NaBO<sub>2</sub>· H<sub>2</sub>O<sub>2</sub> · 3 H<sub>2</sub>O.

**REFERENCES:** 

S. Tanatar. Z. phys. Chem. 26, 132 (1899).

P. Melikoff and L. Pissarjewski. Ber. dtsch. chem. Ges. <u>31</u>, 678 (1898); Z. anorg. allg. Chem. 18, 59 (1898).

# Lindemann Glass

## (Lithium Beryllium Borate)

An intimate mixture of 14.4 g. of  $\text{Li}_2\text{CO}_3$ , 6.44 g. of Be (OH)<sub>2</sub> and 50 g. of H<sub>3</sub>BO<sub>3</sub> (all finely powdered) is fused in a Pt crucible until no further CO<sub>2</sub> is evolved and the melt has become transparent. The melt is poured into a graphite crucible preheated to 600°C; the crucible is placed in an electric muffle furnace preheated to 600°C and the furnace is allowed to cool to room temperature overnight.

PROPERTIES:

Very transparent to soft x rays. Used in x-ray windows and specimen tubes for powder x-ray work. The melting point is relatively low. Relatively sensitive to moisture and should be stored in a desiccator.

**REFERENCES:** 

A. Schleede. and M. Wellmann. Z. Kristallogr. (A) 83, 148 (1932). H. Menzel and S. Sliwinski. Z. anorg. allg. Chem. <u>249</u>, 357 (1942).

# **Boron Phosphate**

**BPO**<sub>4</sub>

 $\begin{array}{rcrcr} H_{3}BO_{3} \ + \ H_{3}PO_{4} \ = \ BPO_{4} \ + \ 3 \ H_{2}O \\ 61.8 \ & 98.0 \ & 105.9 \end{array}$ 

Stoichiometric quantities of pure  $H_3BO_3$  and  $H_3PO_4$  are mixed and heated to  $80-100^{\circ}C$  or, alternatively, a solution containing equivalent amounts of the two acids is evaporated on a water bath. The amorphous product is converted to crystals by heating for 2 hours at 1000°C. The corresponding quantity of  $(NH_4)_3PO_4$  may be substituted for the phosphoric acid.

PROPERTIES:

Tetragonal crystals. Slightly soluble in water, not hygroscopic, insoluble in dilute acids, soluble in strong alkalis, stable at red heat.

REFERENCES:

G. Meyer. Ber. dtsch. chem. Ges. 22, 2919 (1889).

C. Aschmann. Chem. Ztg. 40, 960 (1917).

E. Gruner. Z. anorg. allg. Chem. 219, 181 (1934).

#### **Boron Arsenate**

#### BAsO<sub>4</sub>

Like BPO<sub>4</sub>, boron arsenate is prepared by evaporation of a solution containing stoichiometric quantities of  $H_3BO_3$  and  $As_2O_5$ . The  $As_2O_5$  must be tested for arsenic content before use.

PROPERTIES:

Structure identical to BPO<sub>4</sub>; white, porous mass.

**REFERENCES:** 

E. Gruner. Z. anorg. allg. Chem. 219, 181 (1934).

G. E. Schulze, Naturwiss. 21, 512 (1933).

### **Boron Methoxide**

#### B(OCH<sub>3</sub>)<sub>3</sub>

The formation of  $B(OCH_3)_3$  from  $H_3BO_3$  (or  $B_2O_3$ ) and  $CH_3OH$  may be made to go nearly to completion by using an excess of

methanol and distilling the ester product as an azeotropic mixture (b.p.  $54.6^{\circ}$ C) with CH<sub>3</sub>OH. A three-neck flask, fitted with a mercuryseal stirrer and a reflux condenser, is filled with four moles of CH<sub>3</sub>OH. One mole of B<sub>2</sub>O<sub>3</sub> is gradually added with stirring through the third neck, causing the solution to boil gently. The mixture is refluxed for one hour, after which the reflux condenser is replaced with a downward condenser and the azeotrope is distilled off together with a small excess of methanol, at a boiling temperature of 70°C. (Similarly, two moles of H<sub>3</sub>BO<sub>3</sub> may be reacted with 16 moles of CH<sub>3</sub>OH.) The product is isolated by adding to the distillate 12 g. of LiCl or 46 g. of ZnCl<sub>2</sub> for every 100 g. of azeotrope. The liquid separates into two layers, the upper consisting of 99.5% B(OCH<sub>3</sub>)<sub>3</sub>; the bottom layer contains the salt, the excess alcohol and a small quantity of ester. The alcohol may be recovered by distillation.

#### PROPERTIES:

Moisture-sensitive, colorless liquid. Flammable. B.p. 68.5°C.

**REFERENCES:** 

H. I. Schlesinger, H. C. Brown, D. L. Mayerfield and J. R. Gilbreath. J. Amer. Chem. Soc. <u>75</u>, 213 (1953); cites earlier references.

## Trimethylboron

### **B(CH<sub>3</sub>)**<sub>3</sub>

I. A Grignard solution is prepared from 72 g. of magnesium turnings covered with 250 ml. of absolute n-butyl ether and 285 g. of CH<sub>3</sub>Br dissolved in 500 ml. of n-butyl ether. The reaction is carried out in a one-liter, three-neck flask equipped with reflux condenser, dropping funnel and mercury-seal stirrer. It requires about six hours for completion. Two traps with stopcocks are then attached through ground joints to the reflux condenser. The last trap is protected from the air by a mercury bubbler. The apparatus is flushed with dry, oxygen-free N<sub>2</sub> through a gas inlet tube on the dropping funnel, and a slow stream of the gas is allowed to bubble in throughout the subsequent reaction. The two traps are now immersed in -78°C baths and a solution of 61 g. of BF<sub>3</sub> in 400 ml. of n-butyl either is added dropwise over 4 hours from the funnel. The mixture is then warmed to 70°C and maintained at this temperature for two hours more, allowing the product to condense in the cold traps. The yield of crude product is 44 g. The product is best purified by highvacuum distillation from the  $-78^{\circ}$ C trap into a receiver kept at  $-124^{\circ}C$ .

II. Ethyl ether may be substituted for n-butyl ether in the preparation of the Grignard reagent; the BF<sub>3</sub> is then added in the form of its diethyl etherate or bubbled into the Grignard solution as a gas. The stream of BF<sub>3</sub> gas should not be too fast or it will entrain the  $B(CH_3)_3$ . In this procedure, a wash bottle containing about 5 ml. of concentrated H<sub>2</sub>SO<sub>4</sub> is inserted between the condenser and the traps to absorb any ether present in the product gas. The product is not attacked by cold sulfuric acid.

The yield obtained by this method is somewhat lower than that from method I.

#### **PROPERTIES:**

Colorless gas, characteristic unpleasant odor. B.p.  $-20.2^{\circ}$ C; m.p.  $-161.5^{\circ}$ C (Stock and Zeidler). Ignites and burns with a green flame on exposure to air. Not attacked by water at room temperature. At higher temperatures, reaction with water in a sealed tube yields methyl borates and methane. Absorption by aqueous KOH and NH<sub>3</sub> solutions is violent. Ammonia and its derivatives give addition products with trimethylboron.

REFERENCES:

- I. H. C. Brown. J. Amer. Chem. Soc. 67, 374 (1945).
- II. J. Goubeau and H. J. Becher. Z. anorg. allg. Chem. 268, 1 (1952).

Other methods: A. Stock and F. Zeidler, Ber. dtsch. chem. Ges. 54, 535 (1921); E. Wiberg and W. Ruschmann, Ber. dtsch. chem. Ges. 70, 1583 (1937); C. H. Bamford, D. L. Levi and D. M. Newitt, J. Chem. Soc. (London) <u>1946</u>, 486.

## Triethylboron

#### $B(C_2H_5)_3$

According to Brown,  $B(C_2H_B)_3$  may be prepared analogously to  $B(CH_3)_3$  by dropwise addition of an n-butyl ether solution of  $BF_3$  to a solution of  $C_2H_BMgBr$  in n-butyl ether. When the reaction is complete the product is distilled at  $95^{\circ}C$  under  $N_2$ . In Meerwein's procedure, ethyl ether is used instead of n-butyl ether, but in this case the ether and triethylboron are first distilled off together and then separated by fractional distillation under a nitrogen blanket.

**PROPERTIES:** 

Colorless liquid, spontaneously flammable. B.p.  $95^{\circ}$ C; m.p.  $-92.5^{\circ}$ C. d ( $23^{\circ}$ C) 0. 6931.

REFERENCES:

H. Meerwein. J. prakt. Chem. <u>147</u>, 240 (1937).
H. C. Brown. J. Amer. Chem. Soc. <u>67</u>, 374 (1945).

Other trialkylboron compounds may be synthesized by addition of  $BF_3$  to the ether solutions of the respective alkyl magnesium bromides.

## Trimethylboroxine

(CH<sub>3</sub>BO)<sub>3</sub>

 $\begin{array}{rrrr} B(CH_3)_3 &+& B_2O_3 &=& (CH_3BO)_3\\ 55.9 && 69.6 && 125.5 \end{array}$ 

The compound is prepared by heating  $B(CH_3)_3$  and  $B_2O_3$  together in a sealed tube. The  $B_2O_3$  powder is made by dehydrating  $H_3BO_3$ under vacuum over P<sub>2</sub>O<sub>5</sub>, at 220°C (see p. 787). The very hygroscopic oxide is placed with strict exclusion of moisture in a 200-ml. thick-wall Pyrex tube provided with a ground joint, and a meltingpoint capillary is fastened to the tube just below the joint. The tube is connected to a vacuum pump and immersed in liquid nitrogen, and when a high vacuum has been established, a quantity of B (CH<sub>3</sub>)<sub>3</sub> equivalent to 4.25 g. (0.061 mole) of B<sub>2</sub>O<sub>3</sub> is condensed in the tube. The tube is sealed off, heated to 600°C and kept at this temperature for six hours; in the process the contents turn into a clear, colorless liquid. When the tube has cooled down, the tip is broken under a nitrogen blanket and sealed to a tube leading to the vacuum pump. The tube is evacuated and the contents of the tube are transferred into a -78°C trap. The crude product is purified by removing volatile contaminants at  $-45^{\circ}$ C and then distilling the product from a  $-10^{\circ}$ C trap into a receiver held at  $-78^{\circ}$ C.

#### **PROPERTIES:**

Colorless liquid. M.p.  $-38^{\circ}$ C, b.p. 79.3°C. Combines with water, yielding CH<sub>3</sub>B(OH)<sub>2</sub>. The B and O atoms form a symmetrical six-membered ring.

#### REFERENCES:

J. Goubeau and H. Keller. Z. anorg. allg. Chem. <u>267</u>, 1 (1951). A.B. Burg. J. Amer. Chem. Soc. 62, 2228 (1940).

## Tri-n-Butylboroxine

(**n-C<sub>4</sub>H<sub>9</sub>BO**)<sub>3</sub>

 $3 \text{ n-C}_4 \text{H}_9 \text{B}(\text{OH})_2 = (\text{n-C}_4 \text{H}_9 \text{BO})_3 + 3 \text{H}_2 \text{O}_{305.9}$  251.8

The higher homologs of methylboroxine cannot be prepared from  $B_2O_3$  and the corresponding trialkylborons, because the latter decompose at elevated temperatures. Dehydration of the alkylboronic acids is preferable.

Thus,  $n-C_4H_9B(OH)_2$  (cf. the following preparation) is heated to 120-140 °C in a round-bottom flask connected through a stopcock to a vacuum system. The stopcock is then opened for a short time, allowing most of the gas phase to be removed. The stopcock is closed and the dehydration equilibrium reestablished. By repeated opening and closing of the stopcock it is possible to achieve complete dehydration of the n-butylboronic acid, in accordance with the above equation. Some undecomposed acid is volatilized together with the water vapor.

PROPERTIES:

Colorless liquid. Sensitive to air and moisture. M.p. 259°C (extrapolated).

REFERENCES:

H. C. Mattraw. Ch. E. Erickson and A. W. Laubengayer. J. Amer. Chem. Soc. 78, 4901 (1956); cites earlier references.

## n-Butylboronic Acid

#### n-C<sub>4</sub>H<sub>9</sub>B(OH)<sub>2</sub>

A large number of organoboronic acids have been synthesized by the following procedure, given here for n-butylboronic acid as an example.

A one-liter, three-neck flask is fitted with a mercury-seal stirrer, a low-temperature immersion thermometer and a 500-ml. dropping funnel. An inlet and an outlet for nitrogen are also provided, one at each side neck. The outlet tube terminates in a mercury valve which prevents air from entering the system. A branch line leads nitrogen to the top of the funnel, permitting the addition of the Grignard reagent to be made in an inert atmosphere. The entire apparatus is flushed with dry nitrogen for 20 minutes, and 55 g. of pure  $B(OCH_3)_3$  and then 150 ml. of dry ether are added to the flask from the dropping funnel. A 300-ml. portion of a 1.66 N ether solution of n-butyImagnesium bromide is then placed in the funnel and the nitrogen atmosphere is immediately restored. The

reaction flask is now cooled to -75°C, using a large Dewar flask, and the Grignard solution is added dropwise and at a uniform rate over six hours. The temperature is maintained at -70°C during the addition and a continuous, slow stream of nitrogen is led through the flask. Stirring at -70°C is continued for four hours after the addition of reagent has been completed. The mixture is then allowed to stand overnight without further addition of Dry Ice to the low-temperature bath in order to slowly warm the flask contents to room temperature. The contents of the reaction flask then consist of a solid precipitate and a clear supernatant. The dropping funnel is removed and the precipitate carefully broken up with a glass rod. The funnel is replaced and a solution of 30 ml. of concentrated H<sub>2</sub>SO<sub>4</sub> in 300 ml. of H<sub>2</sub>O is added in drops with stirring. The upper (ether) layer is separated, the aqueous phase is extracted with 50 ml. of ether, and the combined ether fractions are evaporated on a water bath. Toward the end of the evaporation, 20 ml. of water is added and the mixture is again heated on the bath, until no further evaporation occurs on cooling. Crystalline n-butylboronic acid separates and is suction filtered. Purification is by drying over 65% H<sub>2</sub>SO<sub>4</sub> in a desiccator filled with nitrogen, followed by recrystallization from warm toluene. The pure, dry acid is much more air sensitive than the moist, crude product and must be stored under nitrogen.

PROPERTIES:

Colorless, thin monoclinic flakes. The dry compound undergoes autoxidation. Heating causes reversible dehydration; for this reason the compound does not melt sharply. Alkylboronic acids are extremely weak and cannot be titrated with NaOH even when mannitol is added.

**REFERENCES:** 

- H. R. Snyder, J. A. Kuck and J. R. Johnson. J. Amer. Chem. Soc. <u>60</u>, 105 (1938); cites earlier references.
- H. C. Mattraw. Ch. E. Erickson and A. W. Laubengayer. J. Amer. Chem. Soc. 78, 4901 (1956).

## n-Butylboron Difluoride

### $n-C_4H_9BF_2$

One half mole of crude, still moist n-butylboronic acid (cf. preceding preparation) is placed in a 200-ml. flask fitted with a distilling head, a downward condenser and a receiver deeply immersed in a  $-78^{\circ}$ C cooling bath. The receiver is connected to a trap maintained at the same temperature. A wash bottle with concentrated  $H_2SO_4$  serves to keep the apparatus moisture free. A gas inlet tube leading from a BF<sub>3</sub> generator is inserted through the distilling head so that it reaches to the bottom of the flask. This tube is provided with a pressure release valve filled with mercury or concentrated  $H_2SO_4$  (as shown in Fig. 24). A vigorous stream of BF<sub>3</sub> gas is bubbled through the n-C<sub>4</sub>H<sub>9</sub>B (OH)<sub>2</sub>, causing a temperature rise and resulting in liquefaction of the solid. The reaction is prevented from becoming too vigorous by adjustment of the gas flow. When the gas absorption has ceased, heat is applied to the flask and the n-butylboron difluoride still remaining in the mixture distills into the cold receiving flask or into the trap. Finally the contents of the receiver and the trap are combined and fractionated under nitrogen.

Instead of  $n-C_4H_9B(OH)_2$ ,  $(n-C_4H_9BO)_3$  may be treated with BF<sub>3</sub>, in which case  $B_2O_3$  is produced along with the  $n-C_4H_9BF_2$ . The procedure described is general and may be used for the preparation of all alkylboron difluorides. Substitution of BCl<sub>3</sub> for BF<sub>3</sub> yields the alkylboron dichlorides.

PROPERTIES:

Colorless liquid. B.p.  $36.4^{\circ}$ C (742 mm.); d ( $25^{\circ}$ C) 0.851. Not spontaneously flammable, but moisture sensitive.

REFERENCE:

P. A. McCusker and L. J. Glunz. J. Amer. Chem. Soc. 77, 4253 (1955).

# Sodium Tetraphenylborate Na[B(C<sub>6</sub>H<sub>5</sub>)<sub>4</sub>]

A mixture of 12.8 g. of magnesium and 80 g. of  $C_6H_5Br$  is reacted in 200 ml, of absolute ether to form a Grignard solution which is filtered (except for the last 5 ml., which is kept for later use) into a three-neck flask equipped with a reflux condenser, a dropping funnel and a stirrer. A solution of 13.2 g. of BF<sub>3</sub> etherate in 50 ml. of ether is then added in drops with vigorous stirring. Before the addition of etherate is complete, the presence of excess Grignard reagent in the solution is verified by the Gilman test (see below). If no reagent is detected, the Grignard solution previously set aside is added drop by drop until the test is positive. An oily layer, which finally solidifies to a yellow cake, separates during the reaction. The ether is removed from the mixture by vacuum distillation at 100°C. The solid residue is suspended in 500 ml. of water, evolving a considerable amount of heat. An approximately equivalent amount of sodium carbonate solution is added to precipitate the magnesium salt and facilitate the subsequent suction
filtration. The filtrate is saturated with NaCl, and the Na  $[B(C_6H_5)_4]$  separates as a white precipitate, which is filtered with suction and washed with half-saturated NaCl solution. The solid product is dried in vacuum over CaCl<sub>2</sub>, ground to a fine powder, and dried again. Further purification is achieved by extracting the tetraphenylborate with anhydrous acetone, concentrating the extract until the solid just begins to separate, adding 100 ml. of chloroform, and distilling until the condensate is pure chloroform. The distillation is then stopped and the precipitate is filtered while still warm and washed with some chloroform. Halogen-free product is thus obtained with a total yield of 60%, based on the BF<sub>3</sub> etherate used.

Gilman test: 0.5-1 ml. of the test solution is reacted at room temperature with an equal volume of a 1% solution of Michler's ketone (di-p-dimethylaminophenyl ketone) in dry benzene. After a few minutes, 1 ml. of water is carefully added with shaking. The resulting solution, which may have to be filtered, is tested with a few drops of a 0.2% solution of iodine in acetic acid. A characteristic greenish-blue color appears in the presence of Grignard reagent. [H. Gilman and G. Schulze, J. Amer. Chem. Soc. <u>47</u>, 2002 (1925).]

PROPERTIES:

White, finely crystalline, very stable compound. Analytical reagent for  $K^+$ .

**REFERENCE:** 

G. Witting and P. Raff. Liebigs Ann. Chem. 573, 195 (1950).

# **SECTION 15**

Aluminum

H. J. BECHER

#### Lithium Aluminum Hydride

#### LiAlH<sub>4</sub>

I.

For the reaction between LiH and  $AlCl_3$  to occur to a noticeable extent, the thick, hard lumps of lithium hydride must be ground to a very fine powder and then sieved. This treatment is performed under nitrogen. A very reactive ether suspension of LiH is obtained if the material is wet-ground with anhydrous ether in a special ball mill (K. Ziegler et al.) and used immediately.

A fresh suspension of 23.5 g. (2.96 moles) of LiH in 200 ml. of ether is introduced (with exclusion of moisture) into a threeneck flask provided with a dropping funnel, a reflux condenser and a stirrer. A solution of 71.2 g. (0.534 mole) of AlCl<sub>3</sub> in 300 ml. of ether is then added in drops with vigorous stirring. The reaction starts immediately, as shown by the boiling of the ether. The rate is kept uniform by adjustment of the addition rate. When all the AlCl<sub>3</sub> solution has been added, stirring is continued until the reaction has subsided, and the mixture is left to stand for some time. The contents of the flask are then filtered, using nitrogen pressure, through a fritted glass filter, and the clear filtrate is concentrated at atmospheric pressure until it has a syrupy consistency. The residual ether may be evaporated in vacuum at  $70^{\circ}$ C. The yield of LiAlH<sub>4</sub>, based on AlCl<sub>3</sub>, is about 85%.

Insufficient grinding of the LiH may prevent the reaction from starting spontaneously, or from going to completion, and lead to sudden, explosive bursts of renewed activity; in such cases the reaction is started by adding 200 ml. of ether, followed by a solution of 3 g. of  $\text{LiAlH}_4$  in 30 ml. of ether, to the finely divided LiH. The subsequent procedure is the same as described above. According to Wiberg, a slight quantity of iodine may be used as an initiator instead of the  $\text{LiAlH}_4$ .

II.

 $4 \text{LiH} + \text{AlBr}_{3} = \text{LiAlH}_{4} + 3 \text{LiBr}_{31.8} \\ 266.7 \quad 38.0 \quad 260.5$ 

According to Wiberg, the difficulties involved in the preparation of sufficiently reactive LiH are not encountered if  $AlBr_3$  is used instead of  $AlCl_3$ . In this case, coarse LiH can be used:

An ether solution of  $AlBr_3$  is prepared by adding small portions of the material (total 267 g. = 1 mole) to 750 ml. of extremely pure, dry ether in a flask cooled with ice-salt mixture. A 33-g. portion (4.1 moles) of coarsely ground LiH is placed in a three-neck flask with a reflux condenser, a stirrer and a dropping funnel. The hydride is covered with 250 ml. of ether, and the  $AlBr_3$  solution is allowed to run in; the addition takes one to two hours. The stirred mixture is then heated to the boiling point of ether and kept at this temperature for a few hours. The LiBr and unreacted LiH precipitate on cooling. The solution is rapidly decanted through a layer of glass wool in a funnel. The filtrate obtained consists of an ether solution of LiAlH<sub>4</sub> saturated with LiBr. The solution may be stabilized by storing it over a small quantity of LiH.

#### PROPERTIES:

Colorless solid, stable in dry air at room temperature. The ether solution is used in inorganic and organic chemistry as a reducing and hydrogenating agent.

#### REFERENCES:

- A. E. Finholt, A. C. Bond and H. I. Schlesinger, J. Amer. Chem. Soc. <u>69</u>, 1200 (1947).
- J. Mahe, J. Rollet and A. Willemart, Bull. Soc. Chim. France, Mém. (5), <u>16</u>, 481 (1949).
- K. Ziegler, H. G. Gellert, H. Martin and K. Nagel, Liebigs Ann. Chem. 589, 91 (1954).
- E. Wiberg, R. Bauer, M. Schmidt and R. Usón, Z. Naturforsch. 6 b, 393 (1951).
- E. Wiberg and M. Schmidt, Z. Naturforsch. 7 b, 59 (1952).

## Calcium Aluminum Hydride

#### $Ca(AlH_4)_2$

 $\begin{array}{rrrr} 4\,{\rm CaH_2} &+& 2\,{\rm AlCl_3} &=& {\rm Ca}({\rm AlH_4})_2 &+& 3\,{\rm CaCl_2} \\ 168.4 & & 266.8 & 102.1 & 333.0 \end{array}$ 

A freshly prepared solution of AlCl<sub>3</sub> in tetrahydrofuran is added to a suspension of finely ground calcium hydride in the same

#### 15. ALUMINUM

solvent. A 30% excess of  $CaH_2$  should be used. The mixture is refluxed under nitrogen for several hours in a flask provided with a stirrer and a reflux condenser. When the solution has cooled, it is rapidly suction filtered and the solvent is slowly distilled from the filtrate under nitrogen. The solid residue evolves further tetrahydrofuran when heated to 80-90°C in aspirator vacuum. The nearly colorless residue from this treatment still contains about 50-60% tetrahydrofuran, in addition to the Ca(AlH<sub>4</sub>)<sub>2</sub>. The hydride content can be raised to 60% by treatment in high vacuum.

Before the  $CaH_2$  can be used in the reaction, it should be ground for 8-24 hours in a porcelain ball mill. The grinding and subsequent preparation of the tetrahydrofuran suspension are done in the absence of moisture; this ensures sufficient reactivity of the hydride. Before heating the reaction mixture it is recommended that a few crystals of iodine or, better still, a small quantity of a previously prepared tetrahydrofuran solution of  $Ca(AlH_4)_2$  be added. The AlCl<sub>3</sub> should be sublimed before use. At worst, it may be only slightly yellow.

PROPERTIES:

The tetrahydrofuran solution of  $Ca(AlH_4)_2$  is rapidly decomposed by moisture. Even on storage in a closed vessel, considerable decomposition occurs after some weeks.

REFERENCE:

W. Schwab and K. Wintersberger, Z. Naturforsch. 8 b, 690 (1953).

## Polymeric Aluminum Hydride

#### $(AlH_3)_n \cdot x O(C_2H_5)_2$

A solution of 1.137 g. of LiAlH<sub>4</sub> in 30 ml. of ether is placed in a small three-neck flask fitted with a nitrogen inlet tube, a reflux condenser and a dropping funnel. The apparatus is thoroughly flushed with dry nitrogen, and a solution of 1.33 g. of AlCl<sub>3</sub> in 20 ml. of ether is dropped in. The ensuing reaction is vigorous, but falls off rapidly. A filter tube with a sealed-in fritted disk is substituted for the funnel, and the solution is filtered out of the flask. The ether is partly evaporated from the filtrate by slowly raising the temperature to 90°C in high vacuum. The white residue has the composition  $(AlH_3)_n \cdot x O(C_2H_5)_2$ , the ratio of AlH<sub>3</sub> to ether depending on the duration and temperature of evaporation; it ranges from 2.8:1 to 4.8:1. The product cannot be obtained entirely free of ether, since hydrogen slowly splits off at temperatures exceeding 90°C.

Immediately after completion of the reaction, the solute consists of monomeric  $AlH_3$ . After a short while, however, a polymer with a high ether content begins to separate.

**REFERENCES:** 

- A. E. Finholt, A. C. Bond and H. J. Schlesinger, J. Amer. Chem. Soc. 69, 1200 (1947).
- E. Wiberg, H. Graf and R. Usón, Z. anorg. allg. Chem. <u>272</u>, 221 (1953).

# Aluminum Chlorohydride

## $Al_2Cl_3H_3$

 $\begin{array}{rcl} \text{AlH}_3 &+ & \text{AlCl}_3 &= & \text{Al}_2\text{Cl}_3\text{H}_3\\ 30.0 & & 133.4 & & 163.4 \end{array}$ 

A fresh ether solution of monomeric  $AlH_3$  is prepared as described in the previous section and the LiCl precipitate is removed by filtration or centrifugation. A solution of  $AlCl_3$  in a small volume of ether is immediately added to the clear  $AlH_3$  solution. The two reactants ( $AlH_3$  and  $AlCl_3$ ) are added in equimolar amounts. The addition of the  $AlCl_3$  prevents the polymerization of the  $AlH_3$ . The ether can then be completely evaporated from the solution, yielding a water-clear liquid, which distills at  $80^{\circ}C$ without decomposition in high vacuum. It has the composition of  $Al_2Cl_3H_3$ .

SYNONYM:

Trichlorodialane.

PROPERTIES:

The ether solution of the compound behaves chemically as a mixture of solutions of  $AlH_3$  and  $AlCl_3$ . It has the advantage of being more stable than an ether solution of pure  $AlH_3$ .

REFERENCE:

E. Wiberg and M. Schmidt, Z. Naturforsch. 6 b, 333, 458, 459 (1951).

808

#### Aluminum Hydride Trimethylaminate

 $AlH_3 \cdot 2 N(CH_3)_3$ ,  $AlH_3 \cdot N(CH_3)_3$ 

 $AlH_3 \cdot 2 N(CH_3)_3$ 

 $\begin{array}{rrrr} AlH_3 &+& 2 N(CH_3)_3 &=& AlH_3 \cdot 2 N(CH_3)_3 \\ 30.0 & & 118.2 & & 148.2 \end{array}$ 

As in the preceding two preparations, a fresh ethereal solution of monomeric  $AlH_3$  is prepared and its concentration adjusted to 0.5 M. A 15-ml. portion of a 1 M ether solution of anhydrous trimethylamine is added to 15 ml. of the hydride solution. If a slight turbidity results, the mixture should be centrifuged. The clear solution is left to stand for one half hour. The ether is then vacuum distilled, starting at  $-78^{\circ}$ C and slowly raising the temperature to  $+20^{\circ}$ C. The residue, a white powder, is slowly sublimed in high vacuum into a receiving flask cooled with an ice-salt mixture; the sublimation temperature is  $40^{\circ}$ C. The composition of the sublimate is  $AlH_3 \cdot 2 N(CH_3)_3$ .

PROPERTIES:

Colorless crystals. M.p.  $95^{\circ}$ C; decomposes above  $100^{\circ}$ C. Fumes in air and is vigorously hydrolyzed in water. Soluble in tetrahydrofuran, ether and benzene.

 $AlH_3 \cdot N(CH_3)_3$ 

The compound is prepared analogously, but using equimolar quantities of the hydride and the amine. The white powder left in the flask after removal of the ether sublimes between 50 and  $60^{\circ}$ C (1 mm.). The product consists of colorless crystals which have the composition AlH<sub>3</sub> · N(CH<sub>3</sub>)<sub>3</sub>.

PROPERTIES:

M.p. 76°C. Otherwise similar to AlH<sub>3</sub> · 2 N(CH<sub>3</sub>) 3r

**REFERENCE:** 

E. Wiberg, H. Graft and R. Usón, Z. anorg. allg. Chem. <u>272</u>, 221 (1953).

#### Diethylaluminum Bromide

#### $Al(C_2H_5)_2Br$

The reaction is carried out in a one-liter, three-neck flask equipped with a stirrer, a reflux condenser and a dropping funnel, as well as an inlet tube for dry, oxygen-free nitrogen. The flask is filled with 107 g. of turnings obtained from an alloy of 30%magnesium and 70% aluminum. Small quantities of iodine and ethyl bromide are added. After a short time the halide can be observed to react with the metal. The reaction flask is then immersed in an oil bath and a total of 496 g. of ethyl bromide is added dropwise during 2.5 hours. The flask is then heated to  $120-140^{\circ}$ C and kept at this temperature for one hour to ensure completion of the reaction. The product is separated from the magnesium bromide by vacuum distillation, using an oil pump, and is redistilled in vacuum (2 mm.,  $75^{\circ}$ C). All operations must be carried out under nitrogen. The yield is 370 g. of  $Al(C_2H_5)_2Br$ , contaminated with  $2\% AlC_2H_5Br_2$ .

REFERENCE:

A. von Grosse and J. M. Mavity, J. Org. Chem. 5, 106 (1940).

## Triethylaluminum

#### $Al(C_2H_5)_3$

Sodium wire (50 g.) is introduced under nitrogen into a threeneck flask fitted with a stirrer, a reflux condenser and a nitrogen inlet tube. Then 150 g. of  $Al(C_2H_5)_2Br$  is added dropwise. The flask is heated to  $105^{\circ}C$  and a vigorous reaction soon begins. The mixture starts to boil and external cooling is necessary. As soon as the reaction quiets down, an additional 186 g. of  $Al(C_2H_5)_2Br$  is added. The temperature is slowly raised to  $200^{\circ}C$  and maintained there for 10 hours, with constant stirring. The flask is then cooled to room temperature and 2 g. of sodium wire is added to ensure completion of the dehalogenation. Stirring is resumed, and the temperature is again raised to  $150^{\circ}C$  and held for 1.5 hours. The  $Al(C_2H_5)_3$  formed is distilled from the reaction mixture in vacuum (2 mm.). Further purification is achieved by fractionation in a vacuum Podbielniak column; the product boiling between 128 and 130°C and 50 mm. is collected.

PROPERTIES:

Formula weight 114.2. Spontaneously flammable. Immediately hydrolyzed by moisture to Al(OH)<sub>3</sub> and  $C_2H_6$ .

**REFERENCES:** 

A. von Grosse, J. M. Mavity, J. Org. Chem. 5, 106 (1940). Recent pressure processes for aluminumtrialkyls starting with Al, H<sub>2</sub> and olefins, as well as their applications to the lowpressure polymerization of ethylene, are given in K. Ziegler et al., Angew. Chem. 67, 424 (1955).

810

#### Triethylaluminum Etherate

 $Al(C_2H_5)_3 \cdot O(C_2H_5)_2$ 

A 24-g. sample of Electron metal turnings (15.1% Al, the remainder Mg and traces of Si) is covered with 400 ml. of ether in a three-neck flask fitted with a reflux condenser, a stirrer and a dropping funnel; 150 g. of ethyl bromide is added by drops. The reaction may be initiated by the addition of a few crystals of iodine. While the reaction is in progress, the ether should be boiling at a uniform rate. When the metal has completely dissolved, the ether is quickly removed, leaving a solid residue from which the triethylaluminum etherate is distilled at 140-180°C into an ice-cooled receiver. The distillation is carried out slowly under nitrogen at 14-16 mm., using an oil bath. The product is redistilled at 112°C and 16 mm.

PROPERTIES:

Formula weight 188.3. Colorless liquid, sensitive to air and moisture. B.p.  $112^{\circ}C$  (16 mm.).

**REFERENCES:** 

E. Krause and B. Wendt, Ber. dtsch. chem. Ges. 56, 466 (1923).

E. B. Baker and H. S. Sisler, J. Amer. Chem. Soc. 75, 4828 (1953).

#### Diethylaluminum Hydride

#### $Al(C_2H_5)_2H$

$Al(C_2H_5)_2Cl$	+ LiH =	= $Al(C_2H_5)_2H$	+ LiCl
120.6	7.9	128.2	42.5

The ether complex of diethylaluminum chloride is prepared first by adding dropwise one half mole of  $AlCl_3$ , dissolved in some ether, to one mole of  $Al(C_2H_5)_3 \cdot O(C_2H_5)_2$ . When the evolution of heat ceases, the excess solvent is evaporated and the  $Al(C_2H_5)_2Cl$ etherate is distilled in vacuum. A 76-g. portion of this product is diluted with 120 ml. of absolute ether and placed in a 500-ml., three-neck flask which has been flushed with dry nitrogen. A suspension of LiH in ether (62 ml., 7 M in LiH) is prepared in a dropping funnel under a nitrogen blanket and allowed to flow into the flask. Lithium chloride separates and the reaction mixture boils. Toward the end of the reaction, stirring is applied and the mixture is heated to  $40^{\circ}C$  for 15 minutes more. The LiCl rapidly precipitates when the flask is allowed to cool. The clear supernatant is then sampled with a pipette and tested for halogen. If the test is positive, a few more drops of the LiH suspension are added. The supernatant should be just halogen free but should not contain any appreciable amount of excess Li. The mixture is then centrifuged and the clear solution decanted. The precipitate is washed with a small volume of ether, which is then evaporated in an aspirator vacuum, the last traces being removed at 50°C. The residual  $Al(C_2H_5)_2H$  is distilled in high vacuum, holding the flask in a bath maintained at 80°C or lower. It is collected in a receiver cooled to -20°C. The distillate is a water-clear liquid. The entire preparation should be carried out without interruption. The presence of solid deposits during the distillation of the crude product lowers the yield, which otherwise may be as high as 70%.

PROPERTIES:

Air- and moisture-sensitive liquid. B.p.  $55-56^{\circ}C$  ( $10^{-3}$  mm.).

REFERENCE:

K. Ziegler, H. G. Gellert, H. Martin and K. Nagel, Liebigs Ann. Chem. <u>589</u>, 91-121 (1954).

## Aluminum Chloride

AlCl<sub>3</sub>

I. One end of a large diameter Vycor tube (25 to 40 mm.) is inserted through one of two holes bored in a cork closing a wideneck flask. The second hole holds a tube leading to the hood. The cork can be dispensed with if the reaction tube can be made to fit tightly into the neck of a long-neck, round-bottom flask. The reaction tube is placed in an electric furnace. The distance between the hot zone and the receiver flask should not be longer than 8 cm., to prevent plugging of the tube by the sublimate. The cork is protected from the heat by an asbestos wrapping. A porcelain boat containing aluminum turnings or powder is inserted in the reaction tube. Gaseous HCl is then passed through from the end opposite the receiver. The rubber tubing connections should be as short as possible. When the air has been completely displaced by the HCl, the furnace is slowly heated until a white mist begins to appear. The flow of HCl is then increased and the temperature raised to prevent premature condensation of the sublimate. The reaction is then allowed to continue until the aluminum has been completely consumed.

II. Commercial AlCl<sub>3</sub>, which usually contains iron and traces of oxygen due to hydrolysis, may be purified by the following procedure. The crude material is mixed with about 10% by weight of aluminum turnings and the mixture is placed in a sublimation flask with a side arm through which a moderate stream of dry HCl is fed during the sublimation. A short, wide, curved tube connects the sublimation flask with the receiving flask and the entire system is protected against moisture by a drying tube. Further purification is by resublimation in a nitrogen atmosphere, using the receiver of the first step as a sublimation flask; the rest of apparatus is similar.

PROPERTIES:

M.p. 193°C, b.p. 180°C; d (17°C) 2.465, d (200°C, liq.) 1.31. Colorless, transparent, hexagonal plates. Very hygroscopic.

**REFERENCES:** 

- I. F. Stockhausen and L. Gattermann, Ber. dtsch. chem. Ges. 25, 3521 (1892).
- II. D. D. Eley and H. Watts, J. Chem. Soc. (London) 1952, 1914.

#### Aluminum Bromide

#### AlBr<sub>3</sub>

I.

Al	+	$1^{1}/_{2} \operatorname{Br}_{2}$	=	AlBr <sub>3</sub>
27.0		239.7		266.7

Very pure  $AlBr_3$  may be prepared in the apparatus shown in Fig. 244. The bottom of the reaction flask is covered with a layer of glass wool, and the desired quantity of very pure, degreased dry aluminum turnings is placed on top. Nitrogen, dried over  $P_2O_5$ , is introduced at e and escapes at f. When the last traces of moist air have been flushed out, stopcock e is closed and the stream of nitrogen is passed from g to f. Dry bromine is then added from the dropping funnel into flask a. The rate of addition is adjusted so that the heat of reaction maintains the resulting  $AlBr_3$  at a temperature sufficient for refluxing in the middle section of the air condenser. A considerable excess of Al should remain after all the bromine has been added. The reaction flask is then heated until the liquid flowing down from b is colorless. The direction of the nitrogen stream is reversed, a drying tube filled with  $P_2O_5$  is attached at g, and the AlBr<sub>3</sub> is distilled from



Fig. 244. Preparation of aluminum bromide.

PROPERTIES:

a into receiver c. When the distillation is complete, the receiver is disconnected without interrupting the nitrogen flow and immediately closed tightly. II. When colorless AlBr<sub>3</sub> is not required, the procedure described in I may be carried out in a regular distillation flask with a dropping funnel for bromine. The bottom of the flask is again covered with glass wool and the aluminum placed on top. The flask is preheated over an open flame to about 100°C. The dropwise addition of Br<sub>2</sub> is then carried out at such a rate that it is rapidly consumed. The temperature should remain below the boiling point of AlBr<sub>a</sub>. When the addition is complete the product is distilled through a descending tube into a receiver.

Colorless, shiny lamellae. Hydrolyzes in moist air. Reacts violently with water (caution!). Soluble in many organic solvents. M.p. 97.5°C, b.p.  $255^{\circ}$ C; d ( $18^{\circ}$ C) 3.205, d ( $100^{\circ}$ C, liq.) 2.64.

REFERENCES:

D. G. Nicholson, P. K. Winter and H. Fineberg in: F. Audrieth, Inorg. Syntheses, Vol. III, New York-London 1950, p. 33.

W. Biltz and A. Voigt, Z. anorg. allg. Chem. <u>126</u>, 48 (1923).

W. Klemm, W. Tilk and S. von Müllenheim, Z. anorg. allg. Chem. <u>176</u>, 14 (1928).

## Aluminum Iodide

#### Alf3

 $\begin{array}{l} \text{Al} + 1^{1}/_{2} I_{2} = \text{AlI}_{3} \\ \text{27 0} \quad 380.7 \quad 407.7 \end{array}$ 

I. Aluminum turnings and a few crystals of iodine are placed in a  $CO_2$ -filled flask and heated until reaction begins. More  $I_2$  is then added until only a small amount of aluminum remains. The

reaction product is kept a little longer in the molten state and then transferred under nitrogen into an Anschütz distillation flask, where it is slowly distilled in an oil pump vacuum. The pump is protected against iodine vapor by two wash bottles filled with activated charcoal.

II. A 20-g. portion of dried and sublimed iodine is dissolved in 80 ml. of  $CS_2$  and heated with 3.5 g. of sheet aluminum in a flask provided with a reflux condenser. When the reaction is over, the mixture is filtered, the filtrate is heated over a water bath (caution: do not use an open flame!) to distill off most of the  $CS_2$ , and the concentrated residue is cooled. Aluminum iodide crystallizes out. The crystals are suction-filtered, washed with hexane and dried at 100°C. The product retains a slight yellow-brown tinge.

PROPERTIES:

Colorless leaflets if pure. Moisture-sensitive. Decomposes to  $I_2$  and  $Al_2O_3$  on heating in air. Soluble in CS<sub>2</sub>, alcohol and ether. M.p. 191°C, b.p. 382°C; d (17°C) 3.948; d (200°C, liq.) 3.26.

**REFERENCES:** 

- I. W. Biltz and A. Voigt, Z. anorg. allg. Chem. <u>126</u>, 50 (1923).
   W. Klemm, W. Tilk and S. von Müllenheim, Z. anorg. allg. Chem. 176, 15 (1928).
- II. W. Nespital, Z. phys. Chem. (B) 16, 164 (1932).

## Aluminum Chloride Hydrate

 $AlCl_3 \cdot 6 H_2O$ 

At 0°C,  $AlCl_3 \cdot 6H_2O$  is slightly soluble (21 mg./100 ml.) in saturated aqueous HCl and is therefore easily isolated from such a solution. The aluminum is dissolved in concentrated HCl and the solution is transferred into a three-neck flask fitted with a stirrer, an inlet tube for HCl gas and an outlet tube. The flask is cooled to 0°C and HCl gas is introduced into the continuously stirred and cooled solution until it is saturated. The inlet tube should not dip into the solution, since it might become clogged with salt, but sufficient absorption of the HCl is ensured by vigorous stirring. A wash bottle with concentrated  $H_2SO_4$  connected to the outlet of the flask serves to indicate the progress of the saturation. The precipitated hydrate is rapidly suction-filtered and, while still cold, washed with some ether and dried on a clay plate.

REFERENCE:

W. Fischer and W. Seidel, Z. anorg. allg. Chem. 247, 333 (1941).

# Sodium Tetrachloroaluminate

### NaAlCl<sub>4</sub>

 $\begin{array}{rrrr} \mathrm{NaCl} &+& \mathrm{AlCl}_3 &=& \mathrm{NaAlCl}_4 \\ \mathrm{58.4} && \mathrm{133.4} && \mathrm{191.8} \end{array}$ 

The reaction is carried out in a Pyrex vessel (Fig. 245), which should be as compact as possible. The stoichiometric



Fig. 245. Preparation of sodium tetrachloroaluminate. amounts of C.P. NaCl and freshly sublimed  $AlCl_3$  are introduced into the pear-shaped part of the reaction vessel. The filling is done under nitrogen and the vessel should be very clean and dry. A melting point capillary is then affixed underneath the ground joint; the apparatus is evacuated to a high vacuum and torch-sealed. It is then immersed as deeply as possible in an oil bath at 200-240°C. The AlCl<sub>3</sub>, which tends to sublime onto the cool parts of the wall,

is driven back by passing a luminous flame over the condensate spots. When the flask contents have become a nearly clear melt, they are poured hot into the side flask of the vessel. The product solidifies on cooling.

PROPERTIES:

Colorless, crystalline material. M.p. 156°C.

REFERENCES:

- H. Gerding and H. Houtgraaf, Rec. Trav. Chim. Pays-Bas 72, 21 (1953).
- N. C. Baenziger, Acta Cryst. (London) 4, 216 (1951).

## **Tetrachloroaluminic Acid Dietherate**

 $HAlCl_4 \cdot 2O(C_2H_5)_2$ 

 $\begin{array}{rrrr} HCl &+& AlCl_{3} &+& 2 O(C_{2}H_{5})_{2} &=& HAlCl_{4} \cdot 2 O(C_{2}H_{5})_{2} \\ 36.5 & & 133.4 & & 148.2 & & 318.1 \end{array}$ 

Dry HCl gas is passed through absolute ether until a concentrated solution is obtained. Slightly less than the stoichiometric amount of anhydrous  $AlCl_3$  is added with cooling and shaking. A

clear oil separates and slowly forms white crystals when the mixture is cooled to  $-20^{\circ}$ C with stirring. The crystals are filtered under nitrogen on a sintered glass filter and washed with ether. The compound is stable on storage if moisture is excluded.

Analogously, AlBr<sub>3</sub> and HBr may be reacted to form  $HAlBr_4 \cdot 2O(C_2H_5)a$ , which is obtained in the form of an oil.

**REFERENCE:** 

E. Wiberg, M. Schmidt and A. G. Galinos, Angew. Chem. <u>66</u>, 443 (1954).

#### Aluminum Chloride Ammoniate

AlCl<sub>3</sub>  $\cdot$  NH<sub>3</sub>

$$AlCl_3 + NH_3 = AlCl_3 \cdot NH_3$$
  
133.4 17.0 150.4

Thoroughly dried ammonia is passed over freshly sublimed  $AlCl_3$  in a long, 20-40 mm. diameter glass tube. Higher ammoniates are formed as the contents melt. The reaction tube is swept with nitrogen and slowly heated in a furnace from 190°C to 360°C at which temperature the monoammoniate  $AlCl_3 \cdot NH_3$  is obtained. It is purified by vacuum distillation at 2 mm., with the flask on an oil bath at 205-210°C.

**PROPERTIES:** 

Moisture-sensitive, colorless substance. M.p. 130°C.

REFERENCES:

- E. Tiede, M. Thimann and K. Sensse, Ber. dtsch. chem. Ges. <u>61</u>, 1568 (1928).
- H. Gerding and H. Houtgraaf, Rec. Trav. Chim. Pays-Bas 74, 15 (1955).

## Aluminum Chloride-Sulfur Dioxide Adduct

 $AlCl_3 \cdot SO_2$ 

$$\begin{array}{l} \text{AlCl}_3 + \text{SO}_2 = \text{AlCl}_3 \cdot \text{SO}_2 \\ 133.4 & 64.0 & 197.4 \end{array}$$

A combustion tube is charged with  $AlCl_3$ , which should be pure and free of iron. An excess of  $SO_2$  is condensed on the chloride. No moisture should be present during the reaction. The tube is torch-sealed and heated to  $50^{\circ}$ C for about half an hour. The excess SO<sub>2</sub> is then allowed to escape by heating the sealed end of the tube. The residue is a viscous, colorless liquid, which gradually solidifies. Its composition is AlCl<sub>3</sub>·SO<sub>2</sub>.

REFERENCES:

O. Ruff, Ber. dtsch. chem. Ges. <u>35</u>, 4954 (1902). H. Gerding and E. Smit, Z. phys. Chem. (B) 51, 204 (1942).

# Aluminum Chloride-Thionyl Chloride Adduct

## $Al_2Cl_6 \cdot SOCl_2$

 $2 \operatorname{AlCl}_{3} + \operatorname{SOCl}_{2} = \operatorname{Al}_{2}\operatorname{Cl}_{6} \cdot \operatorname{SOCl}_{2}$ 266.7 119.0 385.7

Anhydrous  $AlCl_3$  is dissolved at room temperature in excess  $SOCl_2$ . When large quantities are to be prepared, the solubility of the  $AlCl_3$  can be increased by heating the solution. The excess  $SOCl_2$  is distilled from the red-brown solution and the temperature of the bath in which the flask is immersed is then slowly raised. The first fraction is discarded. The  $Al_2Cl_6 \cdot SOCl_2$  distills over at 212 to  $216^{\circ}C$  in the form of oily droplets which solidify in the receiving flask. The product is purified by fractionation, b.p.  $214-215^{\circ}C$ .

REFERENCE:

H. Hecht, Z. anorg. allg. Chem. 254, 44 (1947).

## Aluminum Chloride-Phosphorus Pentachloride Adduct

AlCl<sub>3</sub> · PCl<sub>5</sub>

 $\begin{array}{rrrr} \text{AlCl}_{3} + \text{PCl}_{5} = \text{AlCl}_{3} \cdot \text{PCl}_{5} \\ 133.3 & 208.3 & 341.6 \end{array}$ 

I. Aluminum chloride and more than the stoichiometric amount of  $PCl_5$  are placed in a thick-wall tube sealed at one end. The tube is evacuated using an oil pump and the other end is sealed. The tube is then heated until the solid turns into a clear melt. After cooling, the tube is opened under nitrogen and the contents are poured into a small distillation flask. Vacuum is applied and the temperature slowly raised to  $300^{\circ}C$  to remove the excess  $PCl_5$ .

818

II. A precipitate of  $AlCl_3 \cdot PCl_5$  is produced on mixing a chloroform solution of  $PCl_5$  and  $AlCl_3$ .

PROPERTIES:

Colorless powder. M.p.  $380^{\circ}$ C ( $383^{\circ}$ C, II). Soluble in nitrobenzene. This solution conducts a current and probably contains the ions AlCl<sub>4</sub> and PCl<sub>4</sub>.

REFERENCES:

- I. W. Fischer and O. Jübermann, Z. anorg. allg. Chem. 235, 345 (1938).
- II. Ya. A. Fialkov and Ya. B. Buryanov, Doklady Akad. Nauk. SSSR 92, 585 (1953).

# Aluminum Bromide-Hydrogen Sulfide Adduct

## AlBr<sub>3</sub> · H<sub>2</sub>S

 $AlBr_3 + H_2S = AlBr_3 \cdot H_2S$ 266,7 34,1 300,8

Aluminum bromide is dissolved in liquid  $H_2S$  with exclusion of moisture. When solution is complete, the excess  $H_2S$  is evaporated. Alternatively, dry  $H_2S$  gas may be introduced into a solution of AlBr<sub>3</sub> in CS<sub>2</sub> and the precipitated crystals filtered with exclusion of moisture.

PROPERTIES:

Colorless, moisture-sensitive crystals. M.p. 103°C in a closed tube. Structure: Al tetrahedrally surrounded by 3 Br and 1 S.

REFERENCES:

W. Biltz and E. Keunecke, Z. anorg. allg. Chem. <u>147</u>, 174 (1925).
J. Jakubsohn, Z. phys. Chem. <u>118</u>, 32 (1925).

Armin Weiss, P. Plass and Alarich Weiss, Z. anorg. allg. Chem. 283, 390 (1956).

## Aluminum Iodide-Hexaammoniate

## $AlI_3 \cdot 6 NH_3$

 $\begin{array}{l} \text{AlI}_3 + 6 \,\text{NH}_3 = \text{AlI}_3 \cdot 6 \,\text{NH}_3 \\ \text{407.7} \quad 102.2 \quad 509.9 \end{array}$ 

Approximately 1 g. of coarsely ground, pure  $AlI_3$  is placed in a tared dry flask and weighed. A slow nitrogen stream is passed through a trap containing dry, liquid ammonia at  $-40^{\circ}$ C, and is then introduced into the flask with the AlI<sub>3</sub>, which is kept at  $-70^{\circ}$ C throughout the reaction. The ammonia is absorbed by the iodide, and a fine, white powder forms. Periodic weighing is used as a check for the quantitative absorption of six moles of NH<sub>3</sub> for each mole of AlI<sub>3</sub> present.

If the concentration of ammonia in the nitrogen is too high, or if the cooling is not efficient enough, the reaction may become so vigorous that the iodide will melt and partially sublime. Partial ammonolysis of the  $AlI_3$  may also occur.

**REFERENCE:** 

W. L. Lloyd Taylor, E. Griswold and J. Kleinberg. J. Amer. Chem. Soc. 77, 294 (1955).

## Aluminum Hydroxide

## a-Al(OH)<sub>3</sub> (HYDRARGILLITE)

I. Aluminum hydroxide (C.P. or better) is dissolved with moderate heating in sodium hydroxide solution (saturated in the cold) until saturation is attained. The solution is diluted with water to a density of 1.1, suction-filtered through hard filter paper, and stored for two weeks in a closed vessel. The mother liquor is decanted from the precipitate of Al(OH)<sub>3</sub> formed. The residue is digested for several days with cold water and then washed with hot water until no alkaline reaction is detectable with litmus. The product is dried first over CaCl<sub>2</sub>, then over  $P_2O_5$ . The entire preparation is carried out in Pyrex glassware.

II. Aluminum hydroxide is precipitated from solution by a cold solution of  $(NH_4)_2CO_3$ ; the crystals are thoroughly washed on a leaf filter and added in small portions to a 50% NaOH solution until no further solution occurs. The greater part of the solution is filtered. The remainder, which contains a small quantity of undissolved aluminum hydroxide, is used to seed the filtrate. Well-developed crystals of hydrargyllite form in about eight days. They are then washed free of alkali.

PROPERTIES:

Poor adsorptivity. Converted into boehmite (AlOOH) at 180 to  $200^{\circ}$ C. d ( $20^{\circ}$ C) 2.424.

**REFERENCES:** 

- I. R. Fricke and B. Wullhorst, Z. anorg. allg. Chem. <u>205</u>, 127 (1932).
- II. W. Biltz and K. A. Lehrer, Z. anorg. allg. Chem. <u>172</u>, 299 (1928).

## $\beta$ -**Al(OH)**<sub>3</sub> (BAYERITE)

I. Aluminum hydroxide (C.P. or better) is dissolved with moderate heating in sodium hydroxide solution (saturated in the cold) until saturation is attained. The solution is diluted with water to a density of 1.15 and suction-filtered through hard filter paper. Carbon dioxide is then bubbled through the filtrate for three days. The mother liquor is decanted from the resulting precipitate; the subsequent treatment is identical to that described for hydrargillite (I). The thoroughly washed and dried residue is pure bayerite.

II. Pure aluminum ribbon is cut into strips, degreased and rinsed with freshly distilled acetone. The pieces are covered with a layer of amalgam by a short dip in a 0.1N solution of HgCl<sub>2</sub> and thoroughly washed with distilled water, followed by a final rinsing with double-distilled water. The pieces are then transferred to a Pyrex flask closed off with a soda-lime tube, and covered with double-distilled water. White flakes of bayerite form after a few days.

Initially, the product obtained by procedure II is amorphous according to its x-ray pattern; however, after 25-30 hours, the bayerite lines begin to appear in the powder pattern.

#### PROPERTIES:

Hexagonal crystals. Considered to be a metastable modification of Al(OH)<sub>3</sub>. Industrial Al(OH)<sub>3</sub>, which is made by stirring seeded aluminum hydroxide solutions, is chiefly hydrargillite. Precipitation from acid solution, on the other hand, yields exclusively an amorphous product when carried out in the cold, and chiefly boehmite ( $\alpha$ -AlOOH) when hot solutions are used.

#### **REFERENCES:**

R. Fricke and B. Wullhorst, Z. anorg. allg. Chem. <u>205</u>, 127 (1932). H. Schmäh, Z. Naturforsch. <u>1</u>, 323 (1946).

J. Böhm, Z. anorg. allg. Chem. <u>149</u>, 203 (1925).

R. Fricke, Z. anorg. allg. Chem. 166, 244 (1927); 175, 249 (1928).

## α-AlOOH (BOEHMITE)

I. As described in method II for preparation of bayerite, C.P. aluminum is coated with a layer of amalgam and thoroughly rinsed with double-distilled water in a ground glass flask with a riser tube. When the water is brought to aboil, a violent reaction ensues which, however, dies down before the aluminum has been entirely consumed. The boehmite obtained is separated by decantation and dried over CaCl<sub>2</sub>, followed by drying over  $P_2O_5$ .

II. Aluminum hydroxide precipitated in the cold with ammonia solution is covered with water in an autoclave and heated for two hours at  $200^{\circ}$ C. Pure boehmite is formed.

PROPERTIES:

Boehmite is the chief constituent of bauxite. It is converted into diaspore by heating it in the presence of NaOH for several days at  $350^{\circ}$ C in an autoclave. Natural diaspore is added as a seeding material [V. K. Drushinida, Doklady Akad. Nauk SSSR, N.S., <u>88</u>, 133 (1953)].

**REFERENCE:** 

- I. R. Fricke and K. Jockers, Z. anorg. allg. Chem. <u>262</u>, 3 (1950).
- II. J. Böhm, Z. anorg. allg. Chem. <u>149</u>, 203 (1925).
   H. Ginsberg and M. Köster, Z. anorg. allg. Chem. <u>271</u>, 41 (1952).

## Aluminum Oxide

#### a-Al<sub>2</sub>O<sub>3</sub> (CORUNDUM)

The compound is prepared by heating aluminum hydroxide or  $\alpha$ -Al<sub>2</sub>O<sub>3</sub> to above 1100°C. Diaspore undergoes rapid conversion above 500°C.

PROPERTIES:

Rhombohedral crystals. M.p. 2050°C.

## $\beta$ -Al<sub>2</sub>O<sub>3</sub>

The  $\beta$  form crystallizes from molten aluminum oxide containing up to 5% of an alkali oxide. The transition takes place at a temperature not much below the melting point of cryolite (1020°C) when a mixture of cryolite and  $\alpha$ -Al<sub>2</sub>O<sub>3</sub> is annealed. The  $\alpha$  form is converted into  $\beta$ -Al<sub>2</sub>O<sub>3</sub> after annealing a mixture containing 15-20% cryolite for 20 hours.

PROPERTIES:

Always contains a small amount of alkali. It is therefore occasionally considered to be a very aluminum-rich alkali aluminate. Hexagonal crystals.

REFERENCE:

H. Saalfeld, Z. anorg. allg. Chem. 286, 174 (1956).

822

 $\gamma$ -Al<sub>2</sub>O<sub>3</sub>

Recent investigations have established that annealing of aluminum hydroxides (with the exception of diaspore) at temperatures between 400 and 1000°C yields not only  $\gamma$ -Al<sub>2</sub>O<sub>3</sub>, but a series of various phases ( $\chi$ ,  $\eta$ ,  $\varkappa$ ,  $\delta$ -Al<sub>2</sub>O<sub>3</sub>), all of which probably contain some residual water to stabilize the respective lattices. As far as range of existence and preparative conditions for these modifications is concerned, the original papers should be consulted.

**REFERENCES:** 

- H. C. Stumpf, A. S. Russel and J. W. Newsome, Ind. Eng. Chem. 42, 1398 (1950).
- M. R. Teritan, Compt. Rend. Hebd. Séances Acad. Sci. 230, 1677 (1950).
- Structure Reports 13, 225 (1950).
- H. Thibon, J. Charrier and R. Tertian, Bull. Soc. Chim. France, (Mém) (5) 18, 384 (1951).
- M. K. B. Day and V. J. Hill, Nature (London) 170, 539 (1952).

#### Aluminum Sulfide

#### $Al_2S_3$

I.

Aluminum turnings are placed in a corundum boat inserted in a quartz or hard porcelain tube. One end of the tube is connected to a generator of pure, dry  $H_2S$ ; the other end leads to the hood. The tube is placed in a furnace and  $H_2S$  is passed through. The temperature during the first five hours is 600-630°C, and for the following 12 hours, 1000°C. The product consists of 90-94% aluminum sulfide which still contains some unreacted aluminum and 1-3% oxygen. Higher reaction temperatures lead to a higher oxygen content, because the product reacts with the corundum of the boat.

II.	2  Al +	3S =	Al <sub>2</sub> S <sub>3</sub>
	54.0	96.2	150.2

A stoichiometric mixture of aluminum filings and sulfur powder is placed in a graphite crucible and covered with some excess sulfur. The reaction is started by a short circuit between the bottom of the crucible and a carbon electrode immersed in the crucible. The reaction may also be initiated with a burning magnesium strip. After the reaction is over, the product is placed in a carbon boat or crucible and heated in vacuum for six hours to  $1150^{\circ}$ C to remove excess aluminum. The molten mass is then allowed to cool slowly. Crystals 1-3 mm. thick are obtained. See also the procedure for deuterium sulfide, p. 134.

#### PROPERTIES:

Yellowish powder or crystals; excess aluminum gives the compound a gray tinge. Hydrolyzed by water to  $H_2S$  and  $Al(OH)_3$ .

**REFERENCES:** 

- W. Klemm, K. Geiersberger, B. Schaeler and H. Mindt, Z. anorg. allg. Chem. 255, 288 (1948); cites earlier references.
- J. Flahaut, Compt. Rend. Hebd. Séances Acad. Sci. 232, 334 (1951).

## Aluminum Sulfite

#### BASIC ALUMINUM SULFITE, $Al_2O_3 \cdot 2SO_2 \cdot H_2O$

Sulfur dioxide is introduced into a suspension of Al(OH)<sub>3</sub> in water until a clear solution is obtained. The latter is then placed in a flask closed with a Bunsen valve. The flask is heated to 78-80°C with constant shaking. A magnetic stirrer may be used. Sulfur dioxide is evolved in the process, a slight positive pressure being maintained by the valve. Crystallization sets in suddenly and is essentially complete after four hours. The solid product is filtered and dried under vacuum. It consists of Al<sub>2</sub>O<sub>3</sub> · 2 SO<sub>2</sub> · 3 H<sub>2</sub>O. Heating to 68 to 90°C under vacuum liberates water until the composition Al<sub>2</sub>O<sub>3</sub> · 2 SO<sub>2</sub> · 1 H<sub>2</sub>O is attained.

PROPERTIES:

Formula weight 248.08. Heating above  $100^{\circ}$ C causes decomposition with evolution of SO<sub>2</sub> and water.

**REFERENCE:** 

E. Rosenkranz and G. F. Hüttig, Z. anorg. allg. Chem. 226, 126 (1936).

NEUTRAL ALUMINUM SULFITE, Al<sub>2</sub>O<sub>3</sub> · 3 SO<sub>2</sub> · xH<sub>2</sub>O

A concentrated aqueous solution of aluminum sulfite is prepared according to the directions in the preceding method and evaporated over  $P_2O_5$  in a desiccator. The  $P_2O_5$  should be renewed several times. Sulfur dioxide is passed through the desiccator during the entire drying process, which takes about 10-14 days.

PROPERTIES:

White powder with variable water content.

**REFERENCE:** 

W. Fischer and E. Burger, Z. anorg. allg. Chem. 251, 355 (1943).

## Aluminum Selenide

Al<sub>2</sub>Se<sub>3</sub>

I. Pure, dry aluminum powder (30 g.) is ground together with 50 g. of finely divided, dry precipitated selenium. Then 5 g. of the mixture is transferred into a thick-wall, 200-ml. clay crucible provided with a lid. The crucible is placed in the hood and a small piece of burning magnesium strip is dropped in to ignite the mixture. The crucible is immediately covered again. At intervals of a few seconds the lid is raised a little and about 3 g. of the starting mixture is added until the entire mixture is used up. When the crucible has cooled down, the product is crushed and stored in a closed container.

II. A stoichiometric mixture of aluminum powder and red selenium is placed in a combustion tube, which is evacuated and sealed off. The end of the tube at which the mixture is situated is carefully heated over an open flame until the onset of the reaction, which occurs at the beginning of red heat. The reaction progresses slowly throughout the entire mass and is accompanied by incandescence.

III. Aluminum turnings are placed in a carbon boat inserted in a Pyrex reaction tube. One end of the tube is connected to a source of high vacuum, the other end to a storage tube containing red selenium. The system is evacuated and the selenium is driven into the reaction tube by heating. The reaction tube itself is heated to  $600-650^{\circ}$ C for 36 hours, which is the time required for complete conversion.

PROPERTIES:

Yellow; slowly hydrolyzed in moist air to H<sub>2</sub>Se and Al(OH)<sub>3</sub>.

**REFERENCES:** 

- I. G. R. Waitkins and R. Shutt in: W. C. Fernelius, Inorg. Syntheses, Vol. II, p. 184, New York-London, 1946.
  H. G. Grimm and A. Metzger, Ber. dtsch. chem. Ges. <u>69</u>, 1356 (1936).
- II. O. Glemser and T. Risler, Z. Naturforsch. 3 b, 2 (1948).
- III. W. Klemm et al., Z. anorg. Chem. <u>255</u>, 289 (1948).
   A. Schneider and G. Gattow, Z. anorg. allg. Chem. <u>277</u>, 49 (1954).

## Aluminum Telluride

## Al<sub>2</sub>Te<sub>3</sub>

 $\begin{array}{r} 2 \, \mathrm{Al} \, + \, 3 \, \mathrm{Te} \, = \, \mathrm{Al_2 Te_3} \\ 54.0 \quad 382.8 \quad 436 \, 8 \end{array}$ 

I. A corundum crucible is placed in a quartz tube sealed at one end. A narrow glass tube is inserted deep into the quartz tube, and is used for introduction of a slow stream of nitrogen during the reaction. The Te is placed in the crucible with the help of a small long-stem funnel and then fused into a solid mass by heating the quartz tube. The calculated amount of aluminum powder is then added in small portions. If the usually vigorous reaction does not occur, it may be started by mixing the melt briefly with an iron wire. When the reaction is over, the product is annealed by heating for half an hour at  $800-900^{\circ}C$ .

II. A Pyrex tube (12-mm. diameter) is torch-sealed at one end and is slightly constricted 10 cm. from the sealed end. The lower, sealed part of the tube is charged with tellurium, and a porcelain boat containing somewhat less than the calculated amount of aluminum powder is made to slide down into the tube until it is stopped by the constriction. A second constriction is then made at the open end. The tube is evacuated with an aspirator; the aluminum is heated to a light red glow (caution, the glass may warp) and the tellurium melts. The evaporating tellurium reacts with the aluminum. A small fraction of the tellurium condenses on the colder walls above the aluminum. When all the tellurium has evaporated from the lower end of the tube, the latter is sealed and the condensed tellurium is evaporated from the walls and made to pass several times over the aluminum until its quantity appears to remain constant.

PROPERTIES:

Brown-black, metallic sheen, quite hard; hydrolyzes in moist air.

**REFERENCES:** 

I. W. Klemm et al., Z. anorg. allg. Chem. <u>255</u>, 291 (1947).
 II. L. Moser and K. Ertl, Z. anorg. allg. Chem. <u>118</u>, 272 (1921).

## Aluminum Nitride

#### AlN

I.

$$\begin{array}{rcl} \text{Al} &+ & {}^{1/2}\text{N}_{2} &= & \text{AlN} \\ \text{27.0} & & 14.0 & & 41.0 \end{array}$$

A nickel boat is filled with very pure aluminum powder which has been degreased and dried either by extraction with ether or by heating to  $150^{\circ}$ C in a stream of nitrogen. The boat is placed in a quartz or porcelain tube and heated in an electric furnace while purified nitrogen is passed over it. Even though the nitride starts to form on the surface below  $650^{\circ}$ C, the reaction proper begins only at  $820^{\circ}$ C, when the entire mass begins to glow. At this point the flow of nitrogen should be increased to prevent the N<sub>2</sub> pressure from decreasing owing to the rapid reaction. When the reaction is essentially complete, the mass is allowed to cool in a stream of nitrogen. Since the product still contains some unreacted metal, it is pulverized and reheated under nitrogen for 1-2 hours at  $1100-1200^{\circ}$ C. The product obtained is nearly white and has a nitrogen content not far below theoretical.

II. 
$$Al + NH_3 = AlN + \frac{3}{2}H_2$$
  
27.0 17.0 41.0

To obtain silicon-free AlN, aluminum powder pretreated as above is placed in a trough of molybdenum sheet inside a nickel reaction tube, and  $NH_3$  is led through while the tube is heated to  $1300^{\circ}C$  in an electric furnace.

III. 
$$AlCl_3 \cdot NH_3 = AlN + 3 HCl_{150.4} + 41.0 + 109.4$$

The reaction is performed in the apparatus shown in Fig. 246, which consists essentially of a thick-wall Pyrex tube with an enlargement in the middle and four necks at the top. A thin glass tube (nitrogen inlet) passes through the middle neck and reaches nearly to the bottom. The two side necks contain silver wire leads to a tungsten heating coil w suspended in the reaction tube at the level of the bulb. The fourth neck is an outlet for the gas. The reaction tube is thoroughly dried and AlCl<sub>3</sub> · NH<sub>3</sub>, prepared according to the procedure on p. 817, is placed at the bottom. The nitrogen flow is turned on, coil w is heated to about 1000°C, and



Fig. 246. Preparation of aluminum nitride.
h) heater elements; s) electrical leads for heating coil w.

the  $AlCl_3 \cdot NH_3$  is evaporated at 400°C into the upper chamber of the tube, where it is remelted by the second heating element h and made to flow down again. The decomposition takes place on the tungsten coil, AlN being deposited. When the reaction is complete, the product is scraped off and freed of residual chlorine by heating to incandescence in a nitrogen atmosphere.

PROPERTIES:

Slowly hydrolyzed in moist air. Dry  $O_2$  and HCl attack the compound only above 800°C. M.p. 2150-2200°C; d (25°C) 3.05. Crystallizes in a wurtzite lattice.

**REFERENCES:** 

- I. F. Fichter, Z. anorg. allg. Chem. <u>54</u>, 322 (1907); <u>82</u>, 194 (1912).
   J. Wolf, Z. anorg. allg. Chem. 83,
  - 5. wolf, Z. anorg. alig. Chem. <u>83</u>, 159 (1913); <u>87</u>, 123 (1914).
- II, III: E. Tiede, M. Thimann and K. Sensse, Ber. dtsch. chem. Ges. <u>61</u>, 1568 (1928).

T. Renner, Z. anorg. allg. Chem. 298, 22 (1959).

#### Lithium Aluminum Nitride

Li<sub>3</sub>AlN<sub>2</sub>

 $\begin{array}{rrrr} \text{Li}_{3}\text{N} \ + \ \text{Al} \ + \ ^{1}\!/_{2} \ \text{N}_{2} \ = \ \text{Li}_{3}\text{Al}\text{N}_{2} \\ \text{24.8} \ \ 27.0 \ \ 14.0 \ \ 65.8 \end{array}$ 

A stoichiometric mixture of  $\text{Li}_3\text{N}$  and cleaned and degreased aluminum bronze (99.3% Al) is placed in a molybdenum boat, which is then inserted in an iron tube and heated to 630°C in a stream of nitrogen. A vigorous reaction ensues and the nitrogen is absorbed. The temperature is raised to 750°C for a short time and the product is then cooled under nitrogen.

Alternative procedures: The compound is also formed when AlN is annealed with  $\text{Li}_3N$  or when an alloy of 3 Li and 1 Al is heated in a stream of N<sub>2</sub>.

PROPERTIES:

White powder, hydrolyzed by humid air. Thermally stable up to  $1000^{\circ}$ C. Crystallizes in the cubic system in a CaF<sub>2</sub> super-structure.

REFERENCE:

R. Juza and F. Hund, Z. anorg. allg. Chem. 257, 13 (1948).

#### Aluminum Azide

 $Al(N_3)_3$ 

 $\begin{array}{rrrr} AlH_3 &+& 3 HN_3 &=& Al(N_3)_3 &+& 3 H_2 \\ 30.0 & & 129.1 & & 153.0 \end{array}$ 

An ether solution of  $AlH_3$  is prepared from  $LiAlH_4$  and  $AlCl_3$ , and is then filtered and frozen in a liquid-nitrogen-cooled trap. (cf. p. 807). An excess of anhydrous  $HN_3$  dissolved in ether is added, and the contents of the trap are allowed to melt slowly by gradually removing the Dewar flask. The evolution of hydrogen starts at  $-116^{\circ}C$ . The trap is then allowed to warm to room temperature; vacuum is applied and the ether and excess  $HN_3$  are distilled off with renewed cooling. The product  $Al(N_3)_3$  remains as a white powder.

PROPERTIES:

Very moisture sensitive. The compound may be shock detonated. Soluble in tetrahydrofuran.

**REFERENCE:** 

E. Wiberg and H. Michaud, Z. Naturforsch. 9 b, 495 (1954).

#### Aluminum Phosphide

AlP

Al + P = AlP27.0 31.0 58.0

I. Very pure, finely divided aluminum powder (1.8 g.) and 2.9 g. of purified, dried red phosphorus are ground together in a mortar.

The mixture is placed in a Vycor reaction tube (diameter 20 mm; a little less at the ends), one end of which is connected to a distillation flask containing additional red P, and the other end to a receiving flask. The apparatus is flushed with pure hydrogen. The distillation flask is heated in a continuous stream of  $H_2$  until some phosphorus condenses on the aluminum-phosphorous mixture in the reaction tube. The mixture is then ignited by means of a small but extremely hot flame. The ensuing reaction is short but vigorous. When it is over, the excess P is driven into the receiving flask by heating the entire reaction tube. The tube is then cut at the site of the reaction and the AlP is ground under  $H_2$  and stored in a closed container. The product contains 92-94% AlP.

II. A somewhat less pure product is obtained when a mixture of 27 g. of aluminum powder and 31 g. of red P is ignited with a burning magnesium strip (use safety goggles) in an iron crucible. The crucible should not be more than half full, since the mass expands during the reaction and may overflow. In any case, the crucible should be covered with a lid. The yellowish to gray-black reaction mass is ground in a heated mortar.

#### PROPERTIES:

Yellowish-gray to dark, crystalline. Reacts with water to form  $PH_3$ . Crystallizes in the zinc blende structure.

#### **REFERENCES:**

I. W. E. White and A. H. Bushey, J. Amer. Chem. Soc. <u>66</u>, 1666 (1944).

V. M. Goldschmidt, Ber. dtsch. chem. Ges. <u>60</u>, 1289 (1927).

II. Fonces-Diacon, Compt. Rend. Hebd. Séances Acad. Sci. <u>130</u>, 1314 (1900).

## Lithium Aluminum Phosphide

## $Li_3AlP_2$

First,  $\text{Li}_3\text{Al}$  is prepared by melting together pieces of Li and Al turnings in a 3:1 atomic ratio. The melting process is carried out at 600-700°C under argon. The alloy is ground under CO<sub>2</sub> in an agate mortar and, as was described for AlP, placed in a Vycor tube, one end of which is connected to a distillation flask containing red P. Instead of placing the alloy in direct contact with the tube, it is better to pour it into a boat made of sintered corundum or, preferably,  $ZrO_2$ . The tube is heated in an atmosphere of phosphorus vapor until the reaction starts.

PROPERTIES:

Chemical behavior similar to that of AlP. Crystallizes in a rhombically distorted superstructure of the  $CaF_2$  lattice.

The arsenide  $Li_3AlAs_2$  may be prepared by an entirely analogous procedure; its properties are identical to those of  $Li_3AlP_2$ .

**REFERENCE:** 

R. Juza and W. Schulz, Z. anorg. allg. Chem. 269, 1 (1952).

## Aluminum Orthophosphate

#### AlPO<sub>4</sub>

A concentrated sodium aluminate solution is mixed with concentrated  $H_3PO_4$  until the solution is strongly acidic. It is then transferred into a combustion tube, and the tube is sealed and heated to 250°C for several hours. The white, crystalline product usually contains other phosphates in addition to AlPO<sub>4</sub>. These, however, can be removed because of their solubility in 1:5 aqueous HCl.

PROPERTIES:

Formula weight 121.95. M.p. above  $1460^{\circ}C$ ; d  $(23^{\circ}C)$  2.56. Very slightly soluble in concentrated HCl and HNO<sub>3</sub>. Isomorphous with quartz, the silicon atoms in the lattice being regularly replaced by Al and P atoms. Used for special glasses.

REFERENCES:

H. F. Hüttenlocher, Z. Kristallogr. 90, 509 (1935).

A. D. Schulten, Compt. Rend. Hebd. Séances Acad. Sci. 98, 1583 (1884).

The arsenate  $AlAsO_4$  is isomorphous with  $AlPO_4$ . For the preparation see F. Machatschki and A. Moser, Z. Kristallogr. (A) <u>90</u>, 314 (1935); <u>94</u>, 212 (1936).

#### Aluminum Arsenide

AlAs

 $\begin{array}{rrr} Al + As = AlAs \\ 27.0 & 74.9 & 101.9 \end{array}$ 

Equimolar amounts of the elements are fused together in an evacuated quartz tube at 800°C. Alternatively, the compound may

be prepared in a way similar to that described for AlP, by passing As vapors in a stream of hydrogen over finely divided aluminum powder at about 500°C.

PROPERTIES:

Slowly hydrolyzed by cold water, rapidly by hot water. M.p. above 1200°C. Crystallizes in the zinc blende structure lattice type.

The antimonide AlSb may be prepared in the same manner as AlAs, by fusing the elements in an evacuated quartz vessel.

REFERENCES:

G. Natta and L. Passerini, Gazz. Chim. Ital. <u>58</u>, 458 (1928). V. M. Goldschmidt, Skr. Akad. Oslo 1926, No. 8, 33.

# Aluminum Carbide

## Al<sub>4</sub>C<sub>3</sub>

The purest available aluminum powder is mixed with the stoichiometric quantity of pure, finely divided carbon; the mixture is placed in a carbon crucible sealed with a carbon stopper and heated to  $2000^{\circ}$ C in an atmosphere of H<sub>2</sub>. The heating is discontinued after 30 minutes. The product is orange and contains, in addition to Al<sub>4</sub>C<sub>3</sub>, a small amount of Al metal. The carbide is ground to a powder and the metallic impurity removed by treatment with ice-cold concentrated HCl. The excess carbon floats on the surface and may be skimmed off. The purity of the product is directly related to that of the starting material. The presence of nitrogen results in the formation of nitrogenous compounds.

The conversion may also be achieved by heating in hydrogen for three hours to 1500°C. A bright yellow, microcystalline carbide results.

PROPERTIES:

Golden yellow, hexagonal leaflets. M.p. 2100°C, decomposes above 2200°C. Methane is evolved on hydrolysis.

REFERENCES:

O. Ruff and K. Jellinek, Z. anorg. allg. Chem. 97, 316 (1916).
M. von Stackelberg et al., Z. phys. Chem. (A) <u>175</u>, <u>127</u>, 140 (1936).
L. Wöhler and K. Hofer, Z. anorg. allg. Chem. <u>213</u>, 249 (1933).
E. J. Kohlmeyer and S. Lundquist, Z. anorg. Chem. <u>260</u>, 208 (1949).

# Lithium Aluminum Cyanide LiAl(CN)<sub>4</sub>

 $\begin{array}{rrrr} \text{LiAlH}_4 \ + \ 4 \ \text{HCN} \ = \ \text{LiAl}(\text{CN})_4 \ + \ 4 \ \text{H}_2 \\ 38 \ 0 & 108.1 & 138.1 \end{array}$ 

Anhydrous HCN is condensed in vacuum onto a frozen ether solution of LiAlH<sub>4</sub> and the mixture is allowed to melt slowly. The theoretical amount of H<sub>2</sub> is evolved and LiAl(CN)<sub>4</sub> precipitates. If an excess of HCN is used, it may be removed by evacuation, together with the ether. The LiAl(CN)<sub>4</sub> residue is a white powder.

PROPERTIES:

Hydrolyzes readily. Decomposes after some time even in the absence of oxygen and moisture.

Aluminum cyanide  $Al(CN)_3$  may be prepared in a similar manner, by condensing anhydrous HCN onto a freshly prepared ether solution of monomeric  $AlH_3$ . The product precipitates out with one mole of ether of crystallization. It may be stored for some time in the absence of oxygen and moisture.

**REFERENCE:** 

G. Wittig and H. Bille, Z. Naturforsch. 6 b, 226 (1951).

## Aluminum Methoxide

Al(OCH<sub>3</sub>)<sub>3</sub>

 $\begin{array}{rrrr} AlCl_3 &+& 3\ CH_3OH &+& 3\ NH_3 &=& Al(OCH_3)_3 &+& 3\ NH_4Cl\\ 133.4 && 96.1 && 51.0 && 120.1 && 160.5 \end{array}$ 

A 45-g. portion of freshly sublimed AlCl<sub>3</sub> is dissolved at 0°C in 750 ml. of anhydrous methanol. The solution is allowed to warm up to 5°C and an excess of dry NH<sub>3</sub> is slowly bubbled through the flask. The Al(OCH<sub>3</sub>)<sub>3</sub> precipitate is suction-filtered, washed with methanol and dried over  $P_2O_5$ .

**REFERENCE:** 

S. Teichner, Compt. Rend. Hebd. Séances Acad. Sci. 237, 810 (1953).

## Aluminum Ethoxide

Al(OC<sub>2</sub>H<sub>5</sub>)<sub>3</sub>

I. Aluminum turnings (27 parts) are covered in a round-bottom flask with 276 parts of anhydrous ethanol. Then  $HgCl_2$  (0.2 part) and a trace of iodine are added to start the reaction. The evolution of hydrogen usually begins after a few seconds. If it fails to occur, the flask may be carefully heated on a water bath. If necessary, the aluminum should be slightly etched with dilute NaOH before use and then rinsed with alcohol.

When the reaction slows down, the flask is heated on the water bath for several hours, until the contents become dry and leafy. The excess alcohol is distilled off on an oil bath at  $210-220^{\circ}$ C, and the hot, liquid residue is quickly poured into a Claisen flask with a wide, short air condenser. The ethoxide is distilled at 10 mm. and  $210-220^{\circ}$ C. After a short time, the distillate solidifies to a snow-white mass, which is stored in a well-closed container. The yield is 90%.

II. A 100-g. portion of aluminum turnings is covered with 650 ml. of xylene in a flask equipped with a reflux condenser and a dropping funnel, and the mixture is heated to the boiling point of xylene. Absolute ethanol (440 ml.), containing 0.5 g. of KgCl<sub>2</sub> and a trace of iodine, is added dropwise. The reaction starts immediately, and the heat source may soon be removed. When 320 ml. of ethanol has been added, the reaction slows down and heating is again required. The addition of the alcohol should take about  $1^{3/4}$  hours. Heating is continued somewhat longer, and the mixture is filtered hot through a heated fluted filter. The xylene is completely removed from the filtrate, first by distillation and finally under vacuum. About 400 g. of pure, colorless aluminum ethoxide is left in the flask.

PROPERTIES:

M.p. 130°C, b.p. 210-220°C (10 mm.). Slightly soluble in hot xylene, chlorobenzene and other high-boiling solvents.

**REFERENCES:** 

- I. German patent 286,596.
- II. H. Meerwein and R. Schmidt, Liebigs Ann. Chem. <u>444</u>, 232 (1925).

# Aluminum Triethanolaminate Al(OC<sub>2</sub>H<sub>4</sub>)<sub>3</sub>N

 $\begin{array}{rrrr} Al(OC_3H_7)_3 &+& N(C_2H_4OH)_3 &=& Al(OC_2H_4)_3N &+& 3\,C_3H_7OH \\ 204.2 & 149.2 & 173.2 & 180.3 \end{array}$ 

Aluminum isopropoxide is fused at  $150-160^{\circ}C$  with an equimolar quantity of triethanolamine. The solidified melt is recrystallized from dioxane, yielding an adduct of  $Al(OC_2H_4)_3N$  containing one mole of dioxane. Then the adduct is heated for a considerable time at  $140^{\circ}C$ , dioxane splits off and the solvent-free product is obtained.

PROPERTIES:

Cubic crystals hydrolyzed by water. Soluble in chloroform, benzene and other solvents. Sublimes at above 280°C (13 mm.).

REFERENCE:

F. Hein and P. W. Albert, Z. anorg. allg. Chem. 269, 67 (1952).

#### Aluminum Acetate

#### Al(O<sub>2</sub>CCH<sub>3</sub>)<sub>3</sub>

A 2.81-g. portion of Al ethylate is placed in a small flask equipped with reflux condenser and 15 ml. of acetic anhydride is dropped in. The reaction requires heat. The flask is then heated in an oil bath at  $150-160^{\circ}$ C for another five hours. After the mixture has cooled, a white product precipitates and is then decanted from the liquid phase. The solid residue is dried at 5 mm. and  $100^{\circ}$ C for about three hours. The yield is 3.4 g.

#### PROPERTIES:

Insoluble in benzene. Soluble in water, hydrolyzing to form a gellike precipitate.

**REFERENCE:** 

K. C. Pande and R. C. Mehrotra, Z. anorg. allg. Chem. <u>286</u>, 291 (1956).

## Aluminum Acetylacetonate

 $Al(C_5H_7O_2)_3$ 

 $\begin{array}{rl} Al(NO_3)_3 \ + \ 3 \ CH_3 COCH_2 COCH_3 \ + \ 3 \ NH_3 \ = \ Al(C_5H_7O_2)_3 \ + \ 3 \ NH_4 NO_3 \\ 213.0 & 300 \ 3 & 51.1 & 324.3 \end{array}$ 

I. A small excess of acetylacetone is added to an aqueous solution of  $Al(NO_3)_3$ . Then a dilute solution of  $NH_4OH$  is added very slowly. This procedure gives pure Al acetylacetonate in quantitative yield.

II.  $AlCl_3 + 3 CH_3COCH_2COCH_3 = Al(C_5H_7O_2)_3 + 3 HCl_{133.4} 300.3 324.3$ 

Anhydrous  $AlCl_3$  is dissolved in chloroform, and a slight excess of acetylacetone is added.

PROPERTIES:

Brilliant plaques or prisms, similar to mother of pearl. M.p.  $192-194^{\circ}$ C. Sublimes at  $140^{\circ}$ C (10 mm.). d ( $20^{\circ}$ C) 1.27. Decomposes when heated in air. Insoluble in water; slightly soluble in alcohol, ether and benzene. Can be recrystallized from acetone.

REFERENCE:

G. T. Morgan and H. D. Drew, J. Chem. Soc. (London) <u>119</u>, 1060 (1921).

# SECTION 16

# Gallium, Indium, Thallium E. DÖNGES

## Gallium

## Ga

#### ELECTROLYTIC SEPARATION OF GALLIUM

In the method of Sebba and Pugh, the electrolysis vessel is a 250-ml. beaker. The inside wall of the beaker is lined with a piece of Pt foil 20 cm. long and 3 cm. wide. This serves as the anode. The cathode consists of a thick sheet of Pt, 1.5 cm. wide and 3 cm. long. Its lower edge is wedge shaped and is sealed into the lower end of a glass U tube (Fig. 247a). This seal point is enlarged to a cuplike shape. The cathode is sealed into the U tube in such a fashion that the connection to the conducting Pt wire is just inside the glass-metal seal. (If the wire were to extend beyond this seal, it would rapidly corrode at the point where it comes in contact with the solution. If the apparatus is arranged as indicated, no corrosion can be observed even after several hundred hours of operation.) The other end of the Pt wire dips into a Hg pool which, in turn, is connected to the power supply by another lead wire.

The "cup" formed at the electrode end of the U tube is 3 cm. in diameter and very shallow; nevertheless, it can hold more than 10 g. of molten Ga. As a result of the heat produced by the high resistance of the cell, the Ga separates as the liquid. Because the Ga is in contact with the cathode during operation, it is also cathodic. Hence it can be readily washed, dried and weighed directly in the cup. This arrangement therefore permits quantitative work.

Before loading the cell the Ga is first precipitated as the hydroxide, using no more than the equivalent of 10 g. of the metal. It is then dissolved in the minimum volume of concentrated NaOH and diluted with  $H_2O$  to 150 ml. With a current of 1 amp. (3-4 v.), 6 g. (of the 10) is obtained during the first 24 hours; in the next 24 hours 3.5 g. more separates. The remaining 0.5 g. separates so slowly, however, that it is expedient to precipitate it with sodium hydroxide and to use it in the next electrolysis.

Brauer was also able to obtain good results with a simpler cathode arrangement, shown in Fig. 247 b, when the amount of Ga



Fig. 247. Cathodes for electrolytic separation of gallium. a) according to Sebba and Pugh; b) simple arrangement, which has been used successfully in the University of Freiburg Laboratory. to be separated exceeds 10 g. In some cases aluminum is an impurity and becomes appreciably concentrated in that portion of the Ga which has not been separated by electrolysis. The two are separated by precipitation with cupferron; the precipitate is ashed and converted to the hydroxide by fusion with NaOH.

Residual Ga still adhering to the cathode is removed by rinsing down with warm, dilute hydrochloric acid. The Ga immediately forms tiny spheres, which are readily wiped off.

The separated Ga contains traces of Pt. These can be removed, along with any traces of Pb and Sn which may be present, by the following method: Ga is melted under a layer of water and an equal volume of concentrated hydrochloric acid

added. After the metal has been swirled five minutes underneath the acid, the latter is carefully washed out and some concentrated nitric acid is added. The reaction is vigorous at first and the metal is converted to tiny spheres. The acid is diluted with water after a few minutes and carefully washed out after another 10 minutes. The Ga acquires its metallic luster again on renewed washing with dilute hydrochloric acid (probably because of destruction of a surface oxide film), and the tiny spheres coalesce.

About 5% of the Ga dissolves during purification. Although a mixture of hydrochloric and nitric acids (aqua regia) is not used, the Ga is spectroscopically free of Pt (and also of Pb and Sn) after the treatment.

Gallium can also be recovered from weakly acidic solutions via the method suggested by the Electronics Services Research Laboratory. Thus,  $GaCl_3$  is dissolved in distilled water and electrolyzed in a quartz vessel between a Pt anode and a Ta cathode. The gallium separates at the cathode and sinks to the bottom of the vessel. It is separated, washed, and dried.

PROPERTIES:

Atomic weight 69.72. Lead-gray metal with bluish luster. The melt has a silvery-white luster and a marked tendency to supercool,

but solidifies on seeding with solid Ga or a piece of Dry Ice. Stable in air. M.p. 29.78°C, b.p. 2064°C; d 5.90; d (liq., 30°C) 6.09. Crystal type A 11. Hardness 1.5.

**REFERENCES:** 

F. Sebba and W. Pugh, J. Chem. Soc. (London) <u>1937</u>, 1373.
L. Moser and A. Brukl, Monatsh. Chem. <u>51</u>, <u>325</u> (1929).
G. Brauer, private communication (1950).
Chem. Eng. News <u>34</u>, 2887 (1956).

#### DISSOLVING OF METALLIC GALLIUM IN ACIDS

Gallium is only slightly more noble than Zn. However, it dissolves in mineral acids slowly due to surface passivity phenomena. Hot, concentrated nitric acid is the most effective, dissolving 5 g. of Ga in 10 hours. Sebba and Pugh report achieving rapid solution of Ga in concentrated nitric acid if the metal, which disperses in tiny spheres due to the action of hot acid, is alternately cooled to a powdery acid-metal mixture and then reheated.

#### GALLIUM (III) PERCHLORATE Ga (ClO<sub>4</sub>)<sub>3</sub> · 6 H<sub>2</sub>O

Foster claims that concentrated perchloric acid (especially when mixed with concentrated sulfuric acid) dissolves Ga significantly more rapidly than concentrated nitric acid: 5 g. of Ga, heated in 60 ml. of 72%  $HClO_4$ , dissolves within an hour. The resulting  $Ga(ClO_4)_3 \cdot 6H_2O$  separates almost completely

The resulting  $Ga(ClO_4)_3 \cdot 6H_2O$  separates almost completely on cooling and in such large crystals that it can readily be isolated on a fritted glass filter. The damp crystals should not come in contact with organic materials (e.g., filter paper) since the concentration of the adhering perchloric acid is greater than 72% due to the loss of water in the form of the hexahydrate.

The crystals are vacuum-dried for one hour at 125°C.

PROPERTIES:

Formula weight 476.19. Very readily soluble in water and alcohol; deliquesces in the air. Decomposition begins at  $175^{\circ}C$  at atmospheric pressure and at  $155^{\circ}C$  in vacuum. Crystallizes slowly as large octahedra.

#### **REFERENCES:**

- F. Sebba and W. Pugh, J. Chem. Soc. (London) <u>1937</u>, 1373.
- L. S. Foster, J. Amer. Chem. Soc. <u>61</u>, 3122 (1933).
# Trimethylgallium, Tetramethyldigallane, Digallane

Ga(CH<sub>3</sub>)<sub>3</sub>, Ga<sub>2</sub>H<sub>2</sub>(CH<sub>3</sub>)<sub>4</sub>, Ga<sub>2</sub>H<sub>6</sub>

Wiberg and Johannsen state that  $Ga_2 H_6$  is formed by the reaction of triethylamine and tetramethyldigallane; the latter is formed by electrical glow discharge in a mixture of  $H_2$  and gallium trimethyl.

The work was carried out in a Stock high-vacuum apparatus (see Part I, p. 66) because of the sensitivity of these compounds to air, moisture, and stopcock grease. The apparatus was constructed according to Wiberg and Sutterlin's directions; however, each trap was also connected to the large diameter pump manifold via a Stock valve so that an optimum vacuum could be maintained by the pump during the transport of material.

The fractionation was carried out in a Clusius and Riccoboni apparatus (see also Part I, p. 70), which was incorporated in the high-vacuum system; this apparatus has been simplified by Clusius so that the vapor stream flows through an adjustable ceramic cone valve into the condensation trap.

## A) TRIMETHYL GALLIUM

In the method of Wiberg, Johannsen and Stecher, improved by Coates, 6.75 g. of metallic Ga and 37 g. of  $Hg(CH_3)_2$  are heated to boiling at atmospheric pressure in the presence of a trace of  $HgCl_2$ . A flask with a fused-on 20-cm. fractionating column is used and dry nitrogen is passed through. The column head temperature begins to drop below 92°C [b.p. of  $Hg(CH_3)_2$ ] after two hours and remains at 55-56°C [b.p. of  $Ga(CH_3)_3$ ] after five hours. Small quantities of pure  $Ga(CH_3)_3$  are withdrawn from time to time at the column head over a three-day period, until a residue of about 1 ml. [chiefly  $Ga(CH_3)_3$ ] remains along with the Hg formed. Conversion proceeds almost quantitatively and without decomposition.

The  $Ga(CH_3)_3$  distillate is frozen and transferred under nitrogen to a vacuum apparatus where it can be kept under its own vapor pressure until further use.

### **PROPERTIES:**

Formula weight 114.82. M.p.  $-15.8^{\circ}$ C, b.p. (762 mm.) 55.7; vapor pressure (0°C) 64.5 mm. Very sensitive to oxygen.

### B) TETRAMETHYL DIGALLANE

$$2 \operatorname{Ga}(\operatorname{CH}_{3})_{3} + n \operatorname{H}_{2} = \operatorname{Ga}_{2}\operatorname{H}_{2}(\operatorname{CH}_{3})_{4} + 2 \operatorname{CH}_{4} (\operatorname{C}_{2}\operatorname{H}_{6})$$
229.6 201.6

The principle of the experimental electrolyte-H<sub>2</sub> cell was developed by Stock and Sütterlin and further improved by Wiberg and Johannsen and Wiberg and Stecher. The system is saturated with  $Ga(CH_3)_3$  at 760 mm. and  $-44^{\circ}C$  [the H<sub>2</sub>: $Ga(CH_3)_3$  ratio is then 200:1], placed in an ice-cooled cell, and subjected to a 3.4ma. glow discharge at a cell pressure of 12.2 mm. The hydrogen is freed of O<sub>2</sub> traces by passage over platinized asbestos at 300°C and the resulting moisture removed by freezing out.

The  $Ga_2 H_2(CH_3)_4$  formed is frozen out of the H<sub>2</sub> stream at liquid nitrogen temperature. It is freed of unreacted  $Ga(CH_3)_3$  (condensing at  $-80^{\circ}$ C) and byproduct hydrocarbons (condensing at  $-196^{\circ}$ C) by fractionation at room temperature and condensation at  $-30^{\circ}$ C.

The yield is 60-70%.

PROPERTIES:

Colorless, highly viscous, relatively nonvolatile liquid; solidifies to a glass and softens without any well-defined m.p. Decomposes above 130°C into Ga(CH<sub>3</sub>)<sub>3</sub>, Ga and H<sub>2</sub>. Calculated b.p. 172°C. Vapor pressure (0°C) 0.5 mm.; (20°C) 2 mm.; (95°C) 64 mm. Very sensitive to O<sub>2</sub>, moisture and stopcock grease.

C) DIGALLANE

In the method of Wiberg and Johannsen, 235.1 mg. of tetramethyldigallane and 157.5 mg. of triethylamine, which has been completely dried over pieces of Na and then fractionated, are condensed (3:4 mole ratio) in a cold trap. The trap is sealed and the contents are allowed to thaw. Gentle heating results in conversion to a trimethylgallium-triethylamine adduct and digallane. The reaction products are separated by fractionation at room temperature, in which  $Ga(CH_3)_3 \cdot N(C_2H_5)_3$  (vapor pressure 0.04 mm. at 0°C, m.p. 96°C, b.p. 167°C) is condensed at -10°C and  $Ga_2H_6$  at -196°C.

The yield is 56.5 mg. (theoretical: 56.8 mg.).

PROPERTIES:

Colorless, very mobile liquid. M.p.  $-21.4^{\circ}$ C, vapor pressure (0°C) 2.5 mm.; (54°C) 49.1 mm. Calc. b.p. 139°C, but begins to decompose at 130°C to Ga and H<sub>2</sub>.

**REFERENCES:** 

- E. Wiberg and T. Johannsen, Die Chemie 55, 38 (1942).
- T. Johannsen, Thesis, Munich, 1941.
- G. E. Coates, J. Chem. Soc. (London) 1951, 2011.
- A. Stock, Ber. dtsch. chem. Ges. A <u>54</u>, <u>142</u> (1921); see also Part I, this book, p. 66.
- E. Wiberg and W. Sütterlin, Z. anorg. allg. Chem. 202, 1 (1931).
- K. Clusius and L. Riccoboni, Z. phys. Chem. (B) 38, 81 (1938).
- A. Stock, Z. Elektrochem. 39, 256 (1933).
- E. Wiberg, T. Johannsen and O. Stecher, Z. anorg. allg. Chem. 251, 114 (1943).
- E. Wiberg and O. Stecher, private communication.
- A. Stock and W. Sütterlin, Ber. dtsch. chem. Ges. <u>67</u>, 407 (1934); see also E. Wiberg and M. Schmidt, Z. Naturforsch. <u>7 b</u>, 577 (1952).

## Lithium Tetrahydrogallate

## LiGaH<sub>4</sub>

 $\begin{array}{rl} GaCl_{3} \,+\, 4\,LiH \,=\, 3\,LiCl \,+\, LiGaH_{4} \\ 176.1 & 31.8 & 127.2 & 80.7 \end{array}$ 

Finholt, Bond and Schlesinger report that  $LiGaH_4$  can be prepared in the same manner as  $LiAlH_4$  (see p. 805). A ground joint flask with a curved tube fused on is used. The flask is charged with 470 mg. of LiH (fourfold excess) and is connected to a Stock high-vacuum system by means of a ball joint, which permits the flask to be moved. Then 599 mg. of anhydrous GaCl<sub>3</sub> is sublimed into the reaction flask and about 5 ml. of dry ether condensed on top of it by cooling the flask in liquid nitrogen. The ether is slowly heated until the GaCl<sub>3</sub> dissolves in it. Then the contents are cooled to -80°C.

Lithium hydride is then added slowly to the ether from the side arm by carefully turning the ball joint. After the initial vigorous reaction has subsided, the flask is allowed to warm gradually to room temperature. Wiberg and Schmidt report a quantitative yield if the ether solution is refluxed for 1.5 hours on a water bath  $(35^{\circ}C)$  after the exothermic reaction stage is over. The ether solution of LiGaH<sub>4</sub> is finally forced through a glass frit with dry, CO<sub>2</sub>-free N<sub>2</sub> and separated from excess LiH and precipitated LiCl. Vacuum distillation of the ether at room temperature leaves a white residue of LiGaH<sub>4</sub>.

The final product is analyzed by hydrolysis and measuring the liberated  $H_2$ ; Ga is determined as the hydroxyquinolate. The

yield is 76%, the purity 93%. The yield depends chiefly on the extent to which the LiCl is washed out, and the purity on the extent to which the ether is removed.

Heating decomposes  $LiGaH_{4}$  to Ga and probably LiH and  $H_{2}$ .

REFERENCES:

 A. E. Finholt, A. C. Bond, Jr. and H. J. Schlesinger, J. Amer. Chem. Soc. <u>69</u>, 1199 (1947); see also E. Wiberg and M. Schmidt, Z. Naturforsch. 6 b, 171 (1951); <u>7 b</u>, 576 (1952).

## Gallium(III) Chloride

GaCl<sub>3</sub>

I.

 $Ga_{2}O_{3} + 3 SOCl_{2} = 2 GaCl_{3} + 3 SO_{2}$ 187.4 356.9 352.2 192.2

A mixture of 2.5 g. of ignited  $Ga_2O_3$  and about 8 ml. of  $SOCl_2$ (two- to threefold excess) is heated several hours at 200°C in a sealed tube, according to Hecht, Jander and Schlapmann. Complete chlorination of the  $Ga_2O_3$  occurs. The tube is precooled to -10°C before opening, to reduce the positive pressure generated by the SO<sub>2</sub> formed. The SO<sub>2</sub> is allowed to evaporate at room temperature, excess SOCl<sub>2</sub> is distilled off (b.p. 75.7°C), and GaCl<sub>3</sub> is distilled at 220°C.

Fischer and Jübermann state that  $GaCl_3$  can be obtained completely pure by vacuum sublimation in quartz equipment.

II. Klemm and Tilk obtained  $GaCl_3$  by heating  $Ga_2O_3$  in a stream of  $Cl_2$  and  $CCl_4$ ;  $Cl_2$  alone did not react appreciably with  $Ga_2O_3$  up to 800°C.

III.  $G_a + 1^{1/2} Cl_2 = G_a Cl_3$ 69.7 106.4 176.1

Richards, Craig and Sameshima report formation of  $GaCl_3$  by burning metallic Ga in a stream of  $Cl_2$ . Spectroscopically pure  $GaCl_3$  may be obtained by repeated fractional evaporation in a stream of  $Cl_2$  and  $N_2$  and finally in vacuum.

The following methods was developed by the Electronic Services Research Laboratory for preparation of spectroscopically pure GaCl<sub>3</sub>. Chlorine, mixed with N<sub>2</sub> carrier gas, is bubbled through molten commercial Ga. The resultant GaCl<sub>3</sub> is distilled into a Pyrex tube (e.g., 25 cm. long, 2.5 cm. in diameter) under a N<sub>2</sub> protective atmosphere. The tube is sealed when half full and the GaCl<sub>3</sub> subjected to zone melting. If the starting Ga contains 10-70 mg./kg. of impurities, after 20 passes all impurities are concentrated in one quarter of the preparation. The rest (three quarters) of the ingot so treated consists of a clear, crystalline mass, in which no impurities can be detected by spectroscopic methods. The other quarter, containing the impurities, is unmistakable because of its color and may readily be cut off. The Ga contained in it is recovered by electrolysis (see p. 837). IV. Heyne claims GaCl<sub>3</sub> is best prepared by heating metallic Ga

at 200-400°C in a stream of dry HCl followed by distillation.

### PROPERTIES:

Colorless, needle-shaped, very hygroscopic crystals; fumes and deliquesces in air. M.p. 77.9°C, b.p. 201.3°C; d 2.47. Dissolves in water, evolving large quantities of heat. Very readily soluble in ether; shaking a 5.5N HCl (optimum) GaCl<sub>3</sub> solution with the same volume of ether (repeatedly preextracted with 5.5N HCl) extracts 98% of the GaCl<sub>3</sub> (partition coefficient = 75; see Fig. 248) [E. H. Swift, J. Amer. Chem. Soc. <u>46</u>, 2375 (1924)].



Fig. 248. Partition of Ga(III) chloride between ether and aqueous phase as a function of the HCl concentration in the starting solution.

W. Fischer and S. Lauter [German Patent 801,986 (1949)] claim that other ethers, such as diisopropyl and diisobutyl, are better suited than diethyl ether for the selective extraction from aqueous hydrochloric acid, due to their lower water solubility. Also, their losses would seem to be substantially lower because of their lower solubility, especially in recycle operations. Fischer and Lauter further demonstrated that optimum conditions for the aqueous phase depend chiefly on the chloride ion concentration, rather than on the HCl concentration, and that therefore a considerable portion of the HCl may be replaced by an equivalent quantity of readily soluble metal chloride.

REFERENCES:

- I. H. Hecht, G. Jander and H. Schlapmann, Z. anorg. allg. Chem. <u>254</u>, 255 (1947). W. Fischer and O. Jübermann, Z. anorg. allg. Chem. <u>227</u>, 227 (1936).
- II. W. Klemm and W. Tilk, Z. anorg. allg. Chem. 207, 161 (1932).
- III. T. W. Richards, W. M. Craig and J. Sameshima, Proc. Nat. Acad. Sci. Washington <u>4</u>, 387 (1918); Chem. Eng. News <u>34</u>, 2887 (1956).
- IV. G. Heyne, Thesis, Rostock, 1935.

### Gallium(III) Bromide

GaBr<sub>3</sub>

 $\begin{array}{rrrr} Ga \ + \ 1^{1/_2} \, Br_2 \ = \ GaBr_3 \\ 69.7 \ & 239.8 \ & 309.5 \end{array}$ 

Metallic Ga is heated in a stream of  $N_2$  or  $CO_2$  laden with  $Br_2$  vapor, according to Klemm and Tilk's method. A water-clear melt forms first, becoming yellow to red-brown, due to dissolved  $Br_2$ , when all the Ga has reacted. When the bromination is complete, the GaBr<sub>3</sub> is distilled in an inert,  $Br_2$ -free gas stream into a receiver and thus simultaneously freed of dissolved  $Br_2$ .

Fischer and Jübermann prepared very pure  $GaBr_3$  by vacuum sublimation in quartz equipment.

**PROPERTIES:** 

Colorless, very hygroscopic crystals. M.p. 121.5°C, b.p. 279°C; d 3.69.

REFERENCES:

W. Klemm and W. Tilk, Z. anorg. allg. Chem. <u>207</u>, 161 (1932).
W. Fischer and O. Jübermann, Z. anorg. Chem. 227, 227 (1936).

# Gallium(III) Iodide Gals

 $Ga + 1^{1/2}I_2 = GaI_3$ 69.7 380.8 450.5

Gallium iodide is prepared by Klemm and Tilk's method in the apparatus shown in Fig. 249. Metallic Ga is placed in the center section of the tube and the stoichiometric quantity of  $I_2$  in the right side flask. After evacuation and sealing at  $\alpha$ ,  $I_2$  is sublimed



back and forth over the Ga, heated to the reaction temperature, until the conversion is complete. The gallium reacts with a clear luminous flame.

The iodide is purified by Fischer and Jübermann's method in which it is volatilized in vacuum, using quartz equipment.

**PROPERTIES:** 

Fig. 249. Preparation of gallium(III) iodide.

Pale yellow, hygroscopic crystals, fuming in air. The melt is

orange-brown. M.p. 212°C, b.p. 346°C; d 4.15.

REFERENCES:

W. Klemm and W. Tilk, Z. anorg. allg. Chem. 207, 161 (1932).

- W. Fischer and O. Jübermann, Z. anorg. allg. Chem. 227, 227 (1936).
- I. D. Corbett and R. K. McMullan, J. Amer. Chem. Soc. <u>77</u>, 4217 (1955).

## Gallium (II) Chloride and Gallium (II) Bromide

GaCl<sub>2</sub>, GaBr<sub>2</sub>

GALLIUM (II) CHLORIDE

The compound is prepared according to Miescher and Wehrli, by passing dry HCl over metallic Ga which is gently heated in a Pyrex tube. Partial formation of highly volatile  $GaCl_3$  results. Heating the reaction product with excess Ga metal in a fused, evacuated Pyrex tube gives pure  $GaCl_2$ . Some Ga readily separates on resublimation, due to partial decomposition of  $GaCl_2$  into  $GaCl_3$  and Ga.

**PROPERTIES:** 

Formula weight 140.63. M.p. 170°C. Disproportionates to GaCl<sub>3</sub> and Ga above 200°C.

GALLIUM (II) BROMIDE

Miescher and Wehrli report preparation of  $GaBr_2$ , in a method identical to that used for  $GaCl_2$ , but employing a stream of dry  $CO_2$  saturated with  $Br_2$  instead of HCl.

**REFERENCE:** 

E. Miescher and M. Wehrli, Helv. Phys. Acta 7, 331 (1934).

## Gallium Hydroxide

### Ga(OH)<sub>3</sub>, GaO(OH)

#### GALLIUM HYDROXIDE

Laubengayer and Engle state that crystalline  $Ga(OH)_3$  can be prepared as follows: a  $GaCl_3$  solution is precipitated at 90°C with NH<sub>3</sub> solution, and the hydroxide is washed thoroughly and heated with 6 ml. of water in a 15-ml. autoclave for about 89 hours at about 167°C. This results in a very fine powder with a moisture content corresponding roughly to that for the trihydroxide and a characteristic, metastable structure, which slowly converts on heating (e.g., a total of 166 hours at 170°C) to the diaspore structure of the metahydroxide GaO(OH) (see below).

**PROPERTIES:** 

Formula weight 120.74. The crystalline form is soluble in dilute mineral acids.

### GALLIUM METAHYDROXIDE

The compound GaO(OH), with a diaspore structure, is obtained in slow precipitation of a sodium gallate solution by atmospheric  $CO_2$ , according to Böhm and Kahan.

Milligan and Weiser add just sufficient  $NH_3$  solution to a  $GaCl_3$  solution to redissolve the precipitated hydrated oxide and

allow the solution to stand in a desiccator over concentrated  $H_2SO_4$ . In about two weeks almost all the Ga precipitates as the crystalline metahydroxide. It is washed free of Cl<sup>-</sup> ions by decantation and dried at room temperature.

A hydrated gallium oxide, precipitated at  $30-40^{\circ}$ C (or at  $90^{\circ}$ C) with NH<sub>3</sub> solution by Laubengayer and Engle's method, is washed free of foreign ions and heated with 6 ml. of water in a 15-ml. autoclave for about 100 hours (or even longer) at 110-300°C; this yields the metahydroxide as small but well-defined crystals which can be dried at  $105^{\circ}$ C without decomposition.

Crystalline  $Ga(OH)_3$  may appear as a metastable intermediate at about 170°C (see above).

**PROPERTIES:** 

Formula weight 102.73. The crystalline form dissolves in dilute mineral acids. Diaspore structure (EO<sub>2</sub> structure type). Heating in an autoclave above  $305^{\circ}$ C converts it to  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>.

REFERENCES:

- A. W. Laubengayer and H. R. Engle, J. Amer. Chem. Soc. <u>61</u>, 1210 (1939).
- J. Böhm and G. Kahan, Z. anorg. allg. Chem. 238, 350 (1938).
- W. O. Milligan and H. B. Weiser, J. Amer. Chem. Soc. <u>59</u>, 1670 (1937); see also R. Roy, V. G. Hill and E. F. Osborn, J. Amer. Chem. Soc. 74, 719 (1952).

## Gallium(III) Oxide

 $\alpha$ -Ga<sub>2</sub>O<sub>3</sub>,  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>

 $\alpha$ -GALLIUM (III) OXIDE

A hot, concentrated solution of  $NaHCO_3$  is added to a hot  $GaCl_3$  solution, according to the methods of Goldschmidt, Barth and Lunde and Laubengayer and Engle. It is then boiled until precipitation of hydrated gallium oxide is complete. The initially gelatinous precipitate is washed free of Cl<sup>-</sup> ions with hot water, which causes some deterioration.

The product is air-dried one hour at room temperature and then heated to  $425^{\circ}$ C in a Pt crucible. Over a period of 24 hours, the hydrated oxide crystallizes and can simultaneously lose its water to such an extent that it finally constitutes only about 1.5-4%of the material; however, it often crystallizes rather poorly and still contains about 20% water even after heating for two weeks.

848

Complete thermal dehydration of  $\alpha$ -Ga<sub>2</sub>O<sub>3</sub> is impossible, since the required temperatures convert it to  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>.

PROPERTIES:

Formula weight 187.44. Very slightly soluble in water, slowly reacts with dilute mineral acids. Corundum structure (D51 type). Heating at 600°C in air produces slow monotropic conversion to  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>. On heating in an autoclave under water vapor above 305°C converts to  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>, below 305°C, to GaO(OH) (see above). d 6.44.

 $\beta$  -GALLIUM (III) OXIDE

Klemm and Von Vogel claim that heating in air to temperatures of at least  $1000-1250^{\circ}$ C is necessary to convert washed hydrated oxide to completely anhydrous, well-crystallized  $\beta$ -Ga<sub>2</sub>O<sub>3</sub>.

Laubengayer and Engle state that completely anhydrous  $\beta$ -Ga<sub>2</sub>O<sub>3</sub> can be prepared by autoclave heating of hydrated gallium oxide under water vapor for 74 hours at 420°C.

**PROPERTIES:** 

M.p. 1740°C. Insoluble in dilute and concentrated mineral acids. Monoclinic or rhombic crystals. d 5.88.

**REFERENCES:** 

- V. M. Goldschmidt, T. Barth and G. Lunde, Skr. Akad. Oslo 1925, No. 7, 24.
- A. W. Laubengayer and H. R. Engle, J. Amer. Chem. Soc. <u>61</u>, 1210 (1939).
- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934); see also R. Roy, V. G. Hill and E. F. Osborn, J. Amer. Chem. Soc. <u>74</u>, 719 (1952).

## Gallium(I) Oxide

Ga<sub>2</sub>O

 $Ga_2O_3 + 4Ga = 3Ga_2O_187.4 278.9 466.3$ 

A sample of  $Ga_2O_3$  is triturated with excess fused Ga. The mixture is placed in a corundum boat and heated under vacuum in a quartz apparatus provided with a cold finger, according to the

method of Brukl and Ortner and Klemm and Von Vogel. A trap cooled with Dry Ice must be inserted between pump and apparatus to prevent Hg from condensing on the condenser. Slow heating is necessary to avoid sudden Ga<sub>2</sub>O sublimation (which begins at  $500^{\circ}$ C) and consequent bumping of part of the Ga<sub>2</sub>O<sub>3</sub>-Ga mixture into the condenser.

Klemm and Schnick report that only after repeated vacuum distillation over excess metallic Ga does the product acquire the oxygen content calculated for  $Ga_2O$ .

PROPERTIES:

Formula weight 155.44. Brown-black powder. Noticeable volatilization at 600 °C in an inert gas stream at 1 atm. and at 500 °C in high vacuum. Stable in dry air. Decomposes in vacuum at 700 °C to  $Ga_2O_3$  and Ga.

#### REFERENCES:

A. Brukl and G. Ortner, Z. anorg. allg. Chem. 203, 23 (1931).

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. 219, 45 (1934).
- W. Klemm and J. Schnick, Z. anorg. allg. Chem. 226, 353 (1936).

## Gallium(III) Sulfide

 $Ga_2S_3$ 

I.

Sulfur vapor is passed over metallic Ga heated to  $1200-1300^{\circ}$ C in a stream of N<sub>2</sub>, according to Brukl and Ortner's method; the Ga is converted to yellow Ga<sub>2</sub>S<sub>3</sub>. The sulfur is passed twice to ensure complete reaction, the product being powdered between passes.

Hahn and Klingler believe that the direct synthesis of  $Ga_2S_3$  is simpler with the apparatus of Klemm and Vogel (preparation of GaS; see below). The reaction temperature is 1250 °C in that case.

II. 
$$Ga_2O_3 [2 Ga(OH)_3] + 3 H_2S = Ga_2S_3 + 3 H_2O [6 H_2O]$$
  
187.4 [241.5] 102.2 235.6 54.1 [108.1]

In Klemm and Vogel's method,  $Ga_2O_3$  is heated in a stream of  $H_2S$  (purified by liquefaction with  $CO_2$ ) for 14 hours at  $600-700^{\circ}C$  and finally for four hours at  $1200^{\circ}C$ . Faintly yellow  $Ga_2S_3$  is obtained, the color depending on the particle size.

Hahn and Klingler state that  $Ga_2S_3$  is prepared more rapidly and at lower temperatures by starting with  $Ga(OH)_3$  dried at  $150^{\circ}C$ instead of  $Ga_2O_3$ . Heating for about 12 hours below  $550^{\circ}C$  gives  $\alpha$ -Ga<sub>2</sub>S<sub>3</sub> with zinc blende structure (B3 structure type), the hightemperature modification,  $\beta$ -Ga<sub>2</sub>S<sub>3</sub> with wurtzite structure (B4 structure type), being formed at  $600^{\circ}C$ . The stable modification, a wurtzite superstructure with ordered distribution of metal atoms, forms after heating several days above  $1000^{\circ}C$ .

PROPERTIES:

M.p. (in vacuum) 1255°C. Slow decomposition in air with evolution of H<sub>2</sub>S. d (x-ray)  $\alpha$  form 3.747,  $\beta$  form 3.676. Transition temperature 550-600°C.

REFERENCES:

- A. Brukl and G. Ortner, Naturwiss. 18, 393 (1930).
- Monatsh. Chem. 56, 358 (1930).
- H. Hahn and W. Klingler, Z. anorg. Chem. 259, 135 (1949).
- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. 219, 45 (1934).
- W. Klemm, K. Meisel and H. U. von Vogel, Z. anorg. allg. Chem. 190, 136 (1930).
- H. Hahn and W. Klingler, Z. anorg. allg. Chem. 278, 333 (1955).
- H. Spandau and F. Klanberg, Z. anorg. allg. Chem. 295, 300 (1958).

## Gallium(II) Sulfide

GaS

```
Ga + S = GaS
69.7 32.1 101.8
```

Some Ga is weighed into an 8-mm. quartz tube sealed at one end, and the tube is drawn out to a width of 2 mm. at a distance 5 cm. from the lower end and bent at a right angle, as described by Klemm and Von Vogel. The stoichiometric amount of S for the formation of GaS is placed in the open-end section and the tube sealed under vacuum.

The side containing the S is then heated so as to fill the whole tube with vapor. The Ga on the other side must be heated to reaction temperature with a blast burner, some  $O_2$  being added toward the end of the reaction to raise the temperature.

When the sulfur has reacted completely, the Ga section, which now contains GaS, is cooled with water to condense the balance of the S. The reaction tube is then melted off at the bend and GaS heated for half an hour in a crucible furnace at  $1100^{\circ}$ C to complete the reaction. The GaS is readily removed after opening the tube.

PROPERTIES:

Yellow, leafy material. Stable to water. Can be sublimed at 900-950°C in high vacuum to hexagonal prisms; m.p. 965°C. Hexagonal layer lattice. d (x-ray) 3.916.

**REFERENCES:** 

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934).
- H. Hahn and G. Frank, Z. anorg. allg. Chem. 278, 340 (1955).
- H. Spandau and F. Klanberg, Z. anorg. allg. Chem. 295, 300 (1958).

## Gallium(I) Sulfide

### Ga<sub>2</sub>S

 $GaS + Ga = Ga_2S$ 101.8 69,7 171.5

I.

Direct synthesis from Ga and S is impossible. Thus, excess fused Ga is triturated with GaS, mixed with  $Ga_2S_3$  to promote better wetting, according to the method of Klemm and Von Vogel, developed by Brukl and Ortner. The mixture is placed in a corundum boat inserted in a quartz apparatus, equipped with a cold finger, in high vacuum. The boat contents are heated to 700-720°C. If these temperatures are exceeded, the sublimed  $Ga_2S$  has too low a sulfur content.

II.  $2 \operatorname{Ga} + \operatorname{H}_2 S = \operatorname{Ga}_2 S + \operatorname{H}_2$ 139.4 34.1 171.5 2.0

According to Gastinger,  $Ga_2S$  is obtained by heating metallic Ga at reduced pressure (10 mm.) to  $1000-1200^{\circ}C$  in a stream of  $H_2S$ . Figure 250 shows the apparatus developed for this purpose.

Reaction tube r (40-50 mm. in diameter), condenser k and boat carrier st, as well as the diffusion unit dk, are made of quartz. The apparatus is first filled with argon at Ar, with stopcocks  $h_1$ , and  $h_5$  and  $h_7$  closed and all others open. After the tubular furnace ro is adjusted to the prescribed temperature, boat carrier st is pulled out, and boat s carrying the Ga is placed on it and inserted into the furnace, through which the stream of Ar is passing. Stopcocks  $h_2$ ,  $h_8$  and  $h_3$  are then closed,  $H_2S$  introduced through  $h_7$ , and the unit connected to a vacuum pump by opening  $h_1$ . The



Fig. 250. Apparatus for preparation of  $Ga_2S$  by heating metallic Ga in a stream of  $H_2S$ .

pressure and rate of the H<sub>2</sub>S stream can be adjusted by suitable choice of capillaries  $c_1$  and  $c_2$ . Manometers  $m_1$  and  $m_2$  and flow meter sm serve to check these process parameters. The diffusion unit dk depresses back-diffusion of Ga<sub>2</sub>S so that it precipitates only at the water-cooled condenser k.

When reaction is complete, the apparatus is refilled with Ar by closing stopcocks  $h_7$  and  $h_1$  and opening  $h_5$ , and condenser k with its precipitated Ga<sub>2</sub>S is withdrawn.

#### **PROPERTIES:**

Gray to gray-black. Oxidizes slowly in air with liberation of  $H_2S$ , turning greenish. d 4.18.

**REFERENCES:** 

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934).
- A. Brukl and G. Ortner, Monatsch. Chem. 56, 36 (1930).
- E. Gastinger, Z. Naturforsch. <u>10b</u>, 115 (1955) and detailed private communications to G. Brauer.
- H. Spandau and F. Klanberg, Z. anorg. allg. Chem. 295, 300 (1958).

## Ammonium Gallium(III) Sulfate

 $NH_4Ga(SO_4)_2 \cdot 12H_2O$ 

A GaCl<sub>3</sub> solution is evaporated to dryness with the stoichiometric amount of  $H_2SO_4$  (Feit's procedure) and the sulfate is dissolved in water. Stirring solid  $(NH_4)_2SO_4$  into the solution results in precipitation of the alum as a crystalline powder. Trivial amounts of impurities remain dissolved in the mother liquor.

The alum is purified by recrystallization from water, which should not be too hot in order to avoid formation of basic salts.

SYNONYM:

Ammonium gallium alum.

PROPERTIES:

Formula weight 496.07. Colorless crystals; isomorphous with the corresponding aluminum alum. Solubility at  $25^{\circ}C$ : 1 part in 3.24 parts water; precipitates basic salts on heating. d. 1.777.

REFERENCE:

W. Feit, Angew. Chem. 46, 216 (1933).

## **Gallium Selenide**

Ga<sub>2</sub>Se<sub>3</sub>, GaSe, Ga<sub>2</sub>Se

Ga<sub>2</sub>Se<sub>3</sub> and GaSe

 $2 Ga + 3 Se = Ga_2Se_3, Ga + Se = GaSe$ 139.4 236.9 376.3 69.7 79.0 148.7

According to Klemm and Von Vogel,  $Ga_2Se_3$  and GaSe are prepared by the method given for GaS (see p. 851). The reaction proceeds with bright red incandescence (flashes of flame).

Ga<sub>2</sub>Se

Klemm and Von Vogel state that  $Ga_2Se$  cannot be directly synthesized from the elements. Heating equivalent amounts of Ga and Se via the procedure outlined for the preparation of GaS (see p. 851) gives only higher selenides contaminated with metallic Ga. However, if this intermediate is placed in a corundum boat which is inserted into a quartz reactor equipped with a cold finger and is heated in high vacuum, a sublimate uniform to x-ray analysis, and having the composition Ga<sub>2</sub>Se, is produced.

PROPERTIES:

 $Ga_2Se_3$ : Black aggregate, red when ground. Fairly hard and brittle. M.p. 1020°C, d (x-ray) 5.203. B3 structure type (zinc blende).

GaSe: Dark red-brown, greasy, lustrous leaflets. M.p. 960°C, d 5.03. Hexagonal and rhombohedral modifications.

Ga<sub>2</sub>Se: Formula weight 218.40. Black. d 5.02.

REFERENCES:

W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934).

H. Hahn and W. Klinger, Z. anorg. Chem. 259, 135 (1949).

K. Schubert, E. Dorre and M. Kluge, Z. Metallkunde 46, 216 (1955).

### Gallium Telluride

### Ga<sub>2</sub>Te<sub>3</sub>, GaTe

According to Klemm and Von Vogel,  $Ga_2Te_3$  and GaTe are prepared in the manner described for GaS (see p. 851).

PROPERTIES:

 $Ga_2Te_3$ : Formula weight 522.27. Black, hard and fairly brittle. M.p. 790°C, d 5.57.

GaTe: Formula weight 197.33. Black, soft, greasy lustrous leaflets, easy to grind. M.p. 824°C; d (x-ray) 5.751. B3 structure type (zinc blende).

REFERENCES:

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934).
- H. Hahn and W. Klinger, Z. anorg. Chem. 259, 135 (1949).

### Gallium Nitride

GaN

I.

$$Ga + NH_3 = GaN + 1^{1/2}H_2$$
  
69.7 17.0 83.7 3.0

According to the method of Hahn and Juza, developed by Johnson, Parsons and Crew, GaN is prepared by heating metallic Ga in a corundum boat for two hours at  $1100^{\circ}$ C in a rapid stream of NH<sub>3</sub> which has been dried over Na. The product is then ground and heated similarly for another two hours. Pale gray, finely crystalline GaN results.

II.  $(NH_4)_3GaF_6 + 4 NH_3 = GaN + 6 NH_4F$ 237.8 68.1 83.7 222.2

According to Hahn and Juza,  $(NH_4)_3 GaF_6$  (see p. 228) is heated 10 minutes at 900°C in a stream of  $NH_3$ , analogously to the preparation of InN (see p. 866). Pure, but yellow GaN results.

### PROPERTIES:

Slowly dissolved by hot concentrated  $H_2SO_4$  and hot concentrated NaOH, but not by concentrated HCl,  $HNO_3$  and aqua regia. Stable in air, sublimes undecomposed at 800°C. d 6.10. B4 structure type (wurtzite).

**REFERENCES:** 

- H. Hahn and R. Juza, Z. anorg. allg. Chem. <u>244</u>, 111 (1940); compare also R. Juza and H. Hahn, Z. anorg. allg. Chem. <u>239</u>, 282 (1938); 244, 133 (1940).
- W. C. Johnson, J. B. Parsons and M. C. Crew, J. Phys. Chem. <u>36</u>, 2651 (1932).

## Gallium Nitrate

## Ga(NO<sub>3</sub>)<sub>8</sub>

The pure hydrate  $Ga(NO_3)_3 \cdot 8H_2O_3$ , described by Einecke and earlier authors, is difficult to prepare. However, anhydrous  $Ga(NO_3)_3$  is readily prepared as follows.

Gallium metal or oxide is dissolved in nitric acid. The solution is repeatedly evaporated to a very small volume and rediluted with water several times until the concentrated solution no longer has the odor of the acid. It is then diluted with enough water to furnish 26 g. of Ga per 100 ml. of solution. At this concentration a spongy, crystalline mass of  $Ga(NO_3)_3 \cdot xH_2O$  separates from the viscous solution after 1-2 days. The crystals are filtered by suction and suction-dried in air for a while. The mother liquor still contains a considerable quantity of Ga, due to the high solubility of the salt, and may be reused or worked up. The crystals are dried in a stream of air, first at room temperature and then at 40°C. The dehydration is complete after about two days.

856

**PROPERTIES:** 

White powder. Yields a clear solution with water; very soluble.

REFERENCES:

E. Einecke, Das Gallium, 1937, p. 98.

G. Brauer and U. Sporkert, 1958, unpublished.

## Gallium Phosphide, Arsenide and Antimonide

### GaP, GaAs, GaSb

Ga + P, As, Sb = GaP, GaAs, GaSb 69.7 31.074.9121.8 100.7 144.6 191.5

Goldschmidt reports producing the above three compounds via the following methods.

GaP (orange-yellow, hardness 5) is formed by heating hydrated gallium oxide in a stream of  $H_2$ , saturated with P vapor, at 500°C.

GaAs (dark gray, hardness 4.2) is formed by heating  $Ga_2O_3$  in a stream of  $H_2$  containing As vapor. M.p. 1238°C.

GaBi has not yet been prepared; fusion of equimolar quantities of Ga and Bi at  $600^{\circ}$ C results only in a mixture of the two elements.

All three compounds crystallize in B3 structure type (zinc blende).

**REFERENCES:** 

V. M. Goldschmidt, Skr. Acad. Oslo 1926, No. 8.

W. Köster and B. Thoma, Z. Metallkunde <u>46</u>, 291 (1955).

## Indium

### In

#### PREREFINING OF CRUDE INDIUM

Crude indium obtained, for example, as a byproduct of the New Jersey Zinc refining process contains 2-5% impurities, about 0.8% Pb, 0.5% Zn, 0.5% Sn, 0.01-0.1% Cu, and 0.01-0.05% Fe. Since Zn separates together with the In in the subsequent electrolytic refining, crude indium is first freed of Zn as follows: it is heated at  $800-1000^{\circ}$ C in an iron tube closed at one end and sufficiently long to prevent spray losses. Steam is blown through the melt, by means

of a narrow pipe, for 15 minutes. The zinc volatilizes almost quantitatively as the oxide, and the residual indium contains only 0.008% Zn.

#### ELECTROLYTIC REFINING

Crude indium is comminuted by rolling and cutting, and dissolved in warm, dilute sulfuric acid containing 100-200 g. H<sub>2</sub>SO<sub>4</sub>/ liter to make the electrolyte solution. The temperature, quantity and concentration of the acid are such that some In remains behind with the Pb. Sn and Cu, but hydrolysis of the In is prevented. Platinum foil or gauze serves as the anode and should be arranged parallel to the cathode. A thin piece of In foil, if available, is used as the cathode; otherwise, highly polished Al foil (rinsed with benzine) is employed. Evaporation of the benzine leaves a thin, hazy, oil film which later facilitates removal of separated In. The Al cathode should be somewhat larger than the anode and its edges coated with a thick coat of wax to prevent In from creeping around and separating on the rear side; this would complicate removal of In. Electrolysis is conducted at 20-35°C with a current density of 1 amp./100 cm? Smooth deposition is aided by hourly addition of 1 ml. of 1% gelatin or carpenter's glue solution per liter of solution.

The separated In is finally fused to a button at  $600-800^{\circ}$ C under a mixture of 1:1 KCN and NaOH. It is 99.95% pure.

#### **PROPERTIES:**

Atomic weight 114.76. Silvery white, highly lustrous, very soft metal. Stable only in dry air. M.p. 156.17°C, b.p. above  $1450^{\circ}$ C; d 7.31. A6 structure type. Hardness about 1.2.

**REFERENCES:** 

F. Ensslin, Die Chemie 55, 347 (1942) and supplementary personal communications, 1948. Compare also F. Ensslin, Metall und Erz 37, 401 (1940).

## Indium(III) Chloride

### InCl<sub>3</sub>

I.

$$\frac{\text{In}_2\text{O}_3}{277.5} + \frac{3}{356.9} = 2 \frac{\text{InCl}_3}{442.3} + \frac{3}{192.2}$$

The compound is prepared by the method given for  $GaCl_3$  (see p. 843) according to Hecht, Jander and Schlapmann. A narrower (9 mm. I.D.) tube with greater wall thickness (3 mm.) should be

selected because a higher reaction temperature (300°C) is necessary. The  $InCl_3$  can be sublimed out in pure form after removal of excess SOCl<sub>2</sub> (b.p. 75.7°C).

п.

$$In + 1^{1/2} Cl_2 = InCl_3$$
  
114.8 106.4 221.1

According to Klemm, metallic In is oxidized in  $InCl_3$  in a stream of  $Cl_2$  which has been carefully dried with concentrated  $H_2SO_4$  and  $P_2O_5$ . A very thoroughly dried quartz apparatus, whose components have been fused together or connected by ground joints, is used. Chlorination proceeds rather vigorously at first, with a pale glow, and it proceeds through the mono- and dichloride steps. These compounds are melts. The end product of the chlorination,  $InCl_3$ , sublimes around 600°C as lustrous spangles which appear on the walls of the receiver. The  $InCl_3$  may be purified by subliming in a stream of  $N_2$  (or  $CO_2$ ) to which some  $Cl_2$  has been added. The sublimate is allowed to cool in a stream of inert,  $Cl_2$ -free gas in order to remove excess chlorine.

PROPERTIES:

Sublimation temperature 498°C. d 3.45. Very hygroscopic. Solubility in water at 22°C, 66.11 g.  $InCl_3/100$  g. solution (d 197). Solubility in absolute alcohol at 22°C, 53.2 g  $InCl_3/100$  g. solution (d 1.40).

Concentration of aqueous solutions, which are readily obtainable, for example, by dissolving In metal in hydrochloric acid, produces a crystalline hydrate, e.g.,  $InCl_3 \cdot 4H_2O$ .

**REFERENCES:** 

- I. H. Hecht, G. Jander and H. Schlapmann, Z. anorg. Chem. 254, 255 (1947).
- II. W. Klemm, Z. anorg. allg. Chem. <u>152</u>, 252 (1926); see also G. P. Baxter and C. M. Alter, J. Amer. Chem. Soc. <u>55</u>, 1943 (1933).
   Solutions: F. Ensslin and H. Dreyer, Z. anorg. allg. Chem. <u>249</u>, 119 (1942).
   F. Ensslin, B. Ziemeck and L. DeSchaepdryver, Z. anorg. Chem. 254, 293 (1947).

## Indium(III) Bromide

### InBr<sub>3</sub>

I.

$$In + 1^{1/2} Br_2 = InBr_3$$
  
114.8 239.8 354.5

According to Thiel, metallic In is heated in a fast air-free stream of  $CO_2$ , which is saturated with  $Br_2$  vapor by passage

through a wash bottle containing  $Br_2$  (the bottle should be placed in a warm water bath). A melt of InBr and InBr<sub>2</sub> forms first. This melt is brown at first, but gradually becomes lighter. Finally it turns to solid InBr<sub>3</sub>. The latter can be readily sublimed to form lustrous crystalline spangles.

To obtain a pure product, it is essential to follow Klemm and Dierks' procedure of eliminating hydrocarbon-lubricated stopcocks. Joints which cannot be dispensed with should be lubricated with phosphoric acid.

II. Ensslin concentrated aqueous solutions of  $InBr_3$ , readily prepared from In and hydrobromic acid, to obtain anhydrous  $InBr_3$ above 33°C. Below 33°C, the solid phase consists of hydrates.

### PROPERTIES:

Sublimation temperature  $371^{\circ}$ C. d 4.74. Deliquesces. Solubility in water at 22°C, 84.64 g.  $InBr_3/100$  g. solution (d 2.84). Solubility in absolute alcohol at 24°C, 74.0 g.  $InBr_3/100$  g. solution (d 2.21).

**REFERENCES:** 

I.

- I. A. Thiel, Z. anorg. allg. Chem. <u>40</u>, 317 (1904).
- W. Klemm and F. Dierks, Z. anorg. allg. Chem. 219, 42 (1934). II. F. Ensslin and H. Drever, Z. anorg. allg. Chem. 249, 119
- II. F. Ensslin and H. Dreyer, Z. anorg. allg. Chem. <u>249</u>, 119 (1942).
  F. Ensslin, B. Ziemeck and L. DeSchaepdryver, Z. anorg. Chem. 254, 293 (1947).

## Indium(III) Iodide

InI<sub>3</sub>

In H	$-1^{1/2}$ I <sub>2</sub>	$= InI_3$
114.8	380.8	495.5

In Thiel's method, metallic In is heated at  $150-200^{\circ}$ C in an airfree stream of CO<sub>2</sub> saturated with iodine vapor. When the liquid reaction product turns deep red-brown due to excess I<sub>2</sub>, the latter is removed in a stream of pure CO<sub>2</sub> at 230°C.

Klemm reports purification of  $InI_3$  without decomposition via vacuum distillation.

II. According to Ensslin, anhydrous  $InI_3$  is formed by concentration of its aqueous solutions (e.g., from In and HI). This procedure is only recommended when large amounts of material are available, since the salt is very soluble in  $H_2O$  and thus the losses are high.

**PROPERTIES:** 

Pale-yellow crystals, melting to a dark brown liquid at 210°C. d 4.68. Very hygroscopic. Solubility in water at 22°C, 92.91 g. InI<sub>3</sub>/100 g. solution (d 3.46).

REFERENCES:

- I. A. Thiel, Z. anorg. allg. Chem. <u>40</u>, 305 (1904).
   W. Klemm, Z. anorg. allg. Chem. <u>152</u>, 252 (1926).
- II. F. Ensslin, B. Ziemeck and L. DeSchaepdryver, Z. anorg. Chem. 254, 293 (1947).

## Indium(II) Chloride, Bromide and Iodide

### InCl<sub>2</sub>, InBr<sub>2</sub>, InI<sub>2</sub>

I.

In Klemm and Dierks' method, weighed amounts of the trihalides and the stoichiometric quantity of metallic In are thoroughly fused in an evacuated glass vessel to ensure complete conversion to the corresponding dihalides. The products are purified by vacuum distillation; in the case of  $InI_2$  the apparatus is sealed off after evacuation to prevent changes in product composition due to loss of iodine.

II. Pure  $InCl_2$  can be easily prepared by Alken, Haley and Terry's procedure through heating  $InCl_3$  [preferred to  $In(OH)_3$ ] in a slow stream of  $H_2 + 15\%$  HCl at temperatures below 600°C (faint red glow). Complete absence of moisture and  $O_2$  is not necessary. The product is heated for 15 minutes at somewhat above its melting point while a slow stream of  $N_2$  is passed through to remove HCl. The pale-yellow melt solidifies to a glass on cooling.

PROPERTIES:

 $InCl_2$ : Formula weight 185.67. Hygroscopic. Decomposes in water to  $InCl_3$  and In. M.p. 235°C; decomposes below the m.p., apparently to  $InCl_3$  and InCl; b.p. ~ 570°C; d 3.65. Rhombic crystals.

In  $Br_2$ : Formula weight 274.59. Disproportionates in water to In  $Br_3$  and red In Br; the latter slowly decomposes further to yield In  $Br_3$  and In.

 $InI_2$ : Formula weight 368.60. M.p. ~ 210°C; d 4.71.

REFERENCES:

W. Klemm and F. Dierks, Z. anorg. allg. Chem. <u>219</u>, 42 (1934).
J. K. Alken, J. B. Haley and H. Terry, Trans. Faraday Soc. <u>32</u>, 1617 (1936).

## Indium(I) Chloride, Bromide and Iodide

InCl , InBr , InI

I.

InCl<sub>3</sub>, InBr<sub>3</sub>, InI<sub>3</sub> + 2 In = 3 InCl, 3 InBr, 3 InI 221.1 354.5 495.5 229.5 450.7 584.0 725.0

In Klemm and Dierks' procedure, excess In metal is allowed to react with the corresponding trihalides, and the resultant monohalides are distilled off in vacuum.

PROPERTIES:

InCl: Formula weight 150.22. Two enantiotropic modifications; one, yellow and light-sensitive; the other, red and light-insensitive. A mixture of red and yellow modifications forms when the melt solidifies in the dark. Standing in the dark for eight days results in complete conversion to the yellow form. Transition temperature of the latter to the red form is  $125-135^{\circ}$ C. The yellow modification rapidly turns green-black in light. Water hydrolyzes InCl rather rapidly to InCl<sub>3</sub> and In. M.p.  $225^{\circ}$ C, b.p.  $608^{\circ}$  C; d (both forms) 4.18.

InBr: Formula weight 194.68. Forms a red to red-brown melt, almost black in thicker layers, solidifying to a carmine red mass. Hydrolyzes slowly in the cold, but rapidly when heated, to form  $InBr_3$  and In. M.p. 220°C, b.p. 662°C. d 4.96.

InI: Formula weight 241.68. Brown-red when a compact solid; crimson red when finely ground and when sublimed; the melts are darker, almost black at the b.p. Relatively stable to water. M.p.  $351^{\circ}$ C, b.p.  $715^{\circ}$ C; d 5.32.

**REFERENCE:** 

W. Klemm and F. Dierks, Z. anorg. allg. Chem. 219, 42 (1934).

## Indium Hydroxide

In(OH)<sub>3</sub>

According to Milligan and Weiser, a small excess of  $NH_3$  solution is added at 100°C to an  $InCl_3$  solution. The precipitate is aged

a few hours at  $100^{\circ}$ C by letting it stand under its mother liquor, then washed free of chloride and dried at room temperature.

PROPERTIES:

Formula weight 165.78. The product contains some adsorbed water; it is powdery and differs in structure from  $In_2O_3$ .

The same precipitation at room temperature gives a gelatinous product containing more adsorbed water; however, it exhibits the same crystal structure.

**REFERENCE:** 

W. O. Milligan and H. B. Weiser, J. Amer. Chem. Soc. <u>59</u>, 1670 (1937).

### Indium (III) Oxide

## In<sub>2</sub>O<sub>3</sub>

Thiel and Luckmann report that to obtain pure  $\ln_2O_3$  it is necessary to ignite the hydroxide to constant weight at 850°C, followed by heating 30 minutes at 1000°C in air.

PROPERTIES:

Formula weight 277.52. Yellow material; the lower the ignition temperature, the more soluble in water; hygroscopic when weakly ignited, nonhygroscopic when strongly ignited. M.p. about  $2000^{\circ}$ C; d 7.04. C sesquioxide structure type.

**REFERENCE:** 

A. Thiel and H. Luckmann, Z. anorg. allg. Chem. 172, 353 (1928).

## Indium(I) Oxide

### In<sub>2</sub>O

$$\frac{\text{In}_2\text{O}_3}{277.5} + \frac{2}{4.0} + \frac{2}{245.5} + \frac{2}{36.0} + \frac{2}{36.0}$$

In the method of Klemm and Von Vogel, as developed by Thiel and Luckmann,  $In_2O_3$  is reduced with  $H_2$  at a temperature less than  $400^{\circ}C$  until the reduction product acquires a composition approximating  $\ln_2 O$  (overreduction results in partial conversion to metallic In). The reduction product is then vacuum-sublimed from a corundum boat in quartz apparatus and collected on a cold finger. The temperature must be so selected that the outside surface of the sublimator is no higher than about 750°C, since decomposition of sublimed  $\ln_2 O$  sets in at higher temperatures.

**PROPERTIES:** 

Black, finely crystalline, brittle and fairly hard; virtually nonhygroscopic, stable in cold water; readily soluble in hydrochloric acid, evolving  $H_2$ . Burns to yellow  $In_2O_3$ , when heated in air. d 6.99.

**REFERENCES:** 

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. 219, 45 (1934).
- A. Thiel and H. Luckmann, Z. anorg. allg. Chem. 172, 353 (1928).

## Indium Sulfides

In<sub>2</sub>S<sub>3</sub>, InS, In<sub>2</sub>S

In<sub>2</sub>S<sub>3</sub>

$$\begin{array}{rl} In_2O_3 \,+\, 3\,H_2S \,=\, In_2S_3 \,+\, 3\,H_2O \\ 277.5 & 102.2 & 325.7 & 54.0 \end{array}$$

According to Klemm and Von Vogel,  $\ln_2O_3$  is heated in a stream of  $H_2S$  (purified by liquefaction with Dry Ice), first for five hours at 500°C and then for eight hours at 700°C. The product is bright red  $\ln_2S_3$ , which turns lustrous black after melting.

Hahn and Klingler maintain that this product is the hightemperature modification of  $\beta$ -In<sub>2</sub>S<sub>3</sub>. The low-temperature  $\alpha$ -In<sub>2</sub>S<sub>3</sub> form is obtained by H<sub>3</sub>S precipitation of an In(III) salt solution containing an acetic acid-acetate buffer, followed by vacuum drying of the precipitate over P<sub>2</sub>O<sub>5</sub> at temperatures below 100°C.

**PROPERTIES:** 

M.p.  $1050^{\circ}$ C in an evacuated, sealed tube.  $\alpha$ -In<sub>2</sub>S<sub>3</sub> is facecentered cubic and isostructural with  $\beta$ -Al<sub>2</sub>O<sub>3</sub>; hygroscopic, adds 0.5 mole of H<sub>2</sub>O and releases it below 300°C in vacuum or a current of dry H<sub>2</sub>S; irreversible transition at about 330°C to nonhygroscopic  $\beta$ -In<sub>2</sub>S<sub>3</sub>, which is isostructural with  $\beta$ -Al<sub>2</sub>O<sub>3</sub> (D 5<sub>7</sub> type). d (x-ray) 4.648. InS and In<sub>2</sub>S

 $In + S = InS, In_2S_3 + 4 In = 3 In_2S$ 114.8 32.1 146.8 325.7 459.0 784.7

According to Klemm and Von Vogel, InS is prepared by direct synthesis from In and S, and  $In_2S$  from In and  $In_2S_3$ , using the method described for GaS (see p. 851).

According to Gastinger,  $In_2S$  is prepared from In and  $H_2S$  analogously to the method described for Ga<sub>2</sub>S (see p. 852).

**PROPERTIES:** 

InS: Red-brown. M.p. 692°C; d 5.18.

In<sub>2</sub>S: Formula weight 261.58. Black in massive form, yellow in thin layers. Stable to cold and hot water. M.p. 653°C; d 5.87.

**REFERENCES:** 

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>219</u>, 45 (1934).
- W. Klemm, K. Meisel and H. U. von Vogel, Z. anorg. allg. Chem. 190, 136 (1930).
- H. Hahn and W. Klingler, Z. anorg. Chem. 260, 97 (1949).
- E. Gastinger, Z. Naturforsch. 10b, 115 (1955).
- K. Schubert, E. Dorre and E. Gunzel, Naturwiss. 41, 448 (1954).

### Indium Selenides and Tellurides

In<sub>2</sub>Se<sub>3</sub>, InSe, In<sub>2</sub>Se and In<sub>2</sub>Te<sub>3</sub>, InTe, In<sub>2</sub>Te

According to Klemm and Von Vogel, mono- to trivalent In selenides and tellurides are synthesized from weighed amounts of the elements by the method described for GaS (see p. 851). Synthesis proceeds with incandescence even at the temperature of a Bunsen burner.

In preparation of  $In_2Se$ , a uniform product is obtained only if the melt is quenched. It is preferable to vacuum-sublime the crude product in quartz equipment provided with a cold finger. The telluride  $\ln_2$ Te can be sublimed in this apparatus as readily and with almost as little residue as  $Ga_2S$ . The sublimate corresponds closely to the theoretical composition.

PROPERTIES:

 $In_2Se_3$ : Black, fairly soft, quite soluble in strong acids. M.p. 890°C; d 5.67.

InSe: Black, dull greasy luster, easily ground (loamy). M.p. 660°C; d 5.55.

In<sub>2</sub>Se: Black, fairly soft. d. 6.17.

 $In_2Te_3$ : Black, hard, brittle. M.p. 667° C; d (x-ray) 5.798. B3 type (zinc blende).

InTe: Silver gray when hot, steel blue when cold; fibrous, readily ground; not appreciably soluble in HCl. M.p.  $696^{\circ}$  C; d. 6.29.

In<sub>2</sub>Te: Dark gray, soft, difficult to triturate. d 6.47.

**REFERENCES:** 

- W. Klemm and H. U. von Vogel, Z. anorg. allg. Chem. <u>260</u>, 97 (1949).
- K. Schubert, E. Dorre and M. Kluge, Z. Metallukunde <u>46</u>, 216 (1955).

### Indium Nitride

InN

$$(NH_4)_{3}InF_6 + 4 NH_3 = InN + 6 NH_4F$$
282.9 68.1 128.8 222.2

According to Hahn and Juza, InN is prepared by placing 1 g. of finely powdered  $(NH_4)_3InF_6$  (see p. 228) in a corundum boat and inserting it into the cold zone of a quartz tube heated by an electric furnace. A fast stream of  $NH_3$  (dried over Na) is allowed to flow through the tube. The boat is pulled into the hot zone of the tube when the temperature has reached 630°C. The temperature thereupon drops to 580-600°C and the material is held at this temperature for 15 minutes. The temperature is then decreased over a period of 10 minutes to 520°C and held there for another 10 minutes to quantitatively drive out the byproduct  $NH_4F$ .

Slow heat-up times and longer heating of the  $(NH_4)_3 InF_6$  lead to products low in N.

**PROPERTIES:** 

Black, air-stable powder; dissolved by NaOH and concentrated  $H_2SO_4$  solutions, but not by other mineral acids. d 6.89. B4 structure type (wurtzite).

**REFERENCE**:

H. Hahn and R. Juza, Z. anorg. allg. Chem. <u>244</u>, 111 (1940); see also R. Juza and H. Hahn, Z. anorg. allg. Chem. <u>239</u>, 282 (1938); 244, 133 (1940).

## Indium Phosphide, Arsenide and Antimonide

InP, InAs, InSb

In + P, As, Sb = InP, InAs, InSb 114.8 31,074.9121.8 145.7 189.7 236.5

In the method of Jandelli, stoichiometric quantities of the reactants are heated in vacuum to prepare these compounds. Only In and Sb combine with relative ease, In and P reacting only to the extent of 94-95% even after 350-400 hours at  $700^{\circ}C$ .

All three compounds crystallize with the zinc blende (B3) structure.

**REFERENCE:** 

A. Jandelli, Gazz. Chim. Ital. 71, 58 (1941).

### Thallium

Tl

#### ELECTROLYTIC SEPARATION OF THALLIUM

I. Richards developed the electrolytic method of separation of thallium from  $Tl_2SO_4$ . It is advisable to treat the solution with  $H_2SO_4$  to avoid cathodic separation of Zn and anodic deposition of  $Tl_2O_3$ . Further, the solution should be sufficiently diluted to depress inclusion of the sulfate in the depositing metal and a low current density should be employed so that the separating metal is as crystalline and compact as possible. Short Pt wires serve as the electrodes. These are immersed in the solution only for a length of 1 cm. The anode, sealed in a glass tube, is inserted into

the solution so that its short free end rests on the bottom of the vessel. The cathode is immersed just below the liquid surface. The separated Tl is frequently removed from the cathode with a narrow glass fork, washed under pure, boiled water and stored there until ready for fusion.

Before fusing, the metal is kneaded into a lump under water and dried as thoroughly as possible between filter papers, and the lumps are beaten together. Fusion at  $350-400^{\circ}$ C is effected in a graphite crucible under a layer of oxalic acid or in a stream of H<sub>2</sub>.

It is advisable to protect Tl from surface oxidation by coating it with a layer of paraffin or storing it under glycerol or petroleum. II. Brown and McGlynn report preparation of a good, smooth, cohesive electrolytic deposit of metallic Tl from a thallium perchlorate bath containing peptone as an anodic depolarizer and cresol as a further additive. Current densities of 0.5 to 1.8 amp./  $100 \text{ cm.}^2$  are used.

Thallium perchlorate is readily soluble in water; the solution is a good conductor and does not change on exposure to air. Lower current densities (about 0.5 amp./100 cm.<sup>2</sup>) provide good deposits on addition of only 10 g. of excess  $HClO_4/liter$ . The peptone gives a yellow precipitate, but does not interfere in any way. Higher current densities (0.9-1.8 amp./100 cm.<sup>2</sup>) also yield good deposits when concentrated solutions with up to 60 g. of free  $HClO_4/liter$  are used.

### PROPERTIES:

Atomic weight 204.39. The fresh surface has a bright metallic luster. Immediately turns gray on exposure to air; barely attacked by  $O_2$ -free H<sub>2</sub>O. M.p. 302.5°C, b.p. + 1457°C. At room temperature  $\alpha$ -Tl, A3 structure type, d 11.84, Mohs hardness 1.3. Transition at 232.3°C to  $\beta$ -Tl, A1 structure type, d 11.88 (measured at room temperature);  $\beta$ -Tl can be supercooled for some time at room temperature without transition.

REFERENCES:

- T. W. Richards and C. P. Smyth, J. Amer. Chem. Soc. <u>44</u>, 525 (1922).
- T. W. Richards and J. D. White, J. Amer. Chem. Soc. <u>50</u>, 3292 (1928).
- K. Lins, Sachtleben A. G., Homberg/Niederrhein, in: A. E. van Arkel, Reine Metalle [Pure Metals], Berlin, 1939, p. 470.
- O. W. Brown and A. McGlynn, Metal Ind. <u>32</u>, 570 (1928); see also E. Bertorelle, L. Giuffre and A. Tunesi, Chim. e l'Ind. <u>32</u>, 517 (1950).

### PURIFICATION OF TECHNICAL GRADE THALLIUM

Technical grade Tl may contain all the flue dust elements, particularly Pb, Ni, Cd, Zn and As. The metal is purified by dissolving in warm, dilute  $H_2SO_4$ , diluting a little and filtering off precipitated PbSO<sub>4</sub>. According to Richards and White, the metal can also be dissolved in hot dilute HNO<sub>3</sub>, since this is a better solvent; the solution of the nitrate is converted to  $Tl_2SO_4$  by heating with a slight excess of concentrated  $H_2SO_4$  until  $SO_3$  fumes are evolved, diluting with water and filtering off insoluble PbSO<sub>4</sub>; the small amounts of Tl(III) which form are reduced with  $SO_2$ .

The dissolved  $Tl_2SO_4$  is further purified by repeated precipitation (as TlCl) with hydrochloric acid, followed by dissolving in warm, dilute sulfuric acid. The thallium is finally electrolyzed as described above.

REFERENCES:

T. W. Richards and J. D. White, J. Amer. Chem. Soc. <u>50</u>, 3292 (1928); private communication, Sachtleben A. G., Homberg/ Niederrhein.

### Thallium(I) Chloride, Bromide and Iodide

#### TlCl, TlBr, TlI

### TICI

A boiling dilute (e.g., 2%)  $Tl_2SO_4$  or  $TINO_3$  solution is treated with dilute HCl until no further precipitate forms. The TlCl is decanted from the mother liquor after cooling, washed with distilled water and dried several hours in an oven at  $110-120^{\circ}C$ .

According to Hönigschmid, Birkenbach and Kothe, TlCl can be purified by repeated crystallization from hot water, or distillation in a stream of dry air or dry  $N_2$ , using quartz equipment. Rotation of the flask distributes the distilled TlCl in a thin film, thus preventing rupture of the vessel wall by contraction of the adhering TlCl on cooling.

#### TlBr

According to Hönigschmid and Striebel, TIBr is precipitated from pure dilute HBr, analogously to TICl. It is harder to

#### E. DÖNGES

recrystallize and is therefore digested several times with boiling water and well dried by suction. It is then dried over KOH in a desiccator and kept in the dark to prevent decomposition. Like TlCl, it can be fused and distilled in a stream of dry  $N_2$ .

## TlĨ

In the Cady and Taft method, TII is precipitated from hot solution with the calculated amount of KI and washed and dried at  $110^{\circ}$ C.

## PROPERTIES:

TlCl: Formula weight 239.85. Completely colorless when freshly prepared; becomes violet on exposure to light. M.p.  $430^{\circ}$ C, b.p.  $806^{\circ}$ C. Solubility in water 0.17 g.(0°C); 0.32 g. (20°C); 2.38 g. (100°C) per 100 g. H<sub>2</sub>O. d 7.0. B2 structure type.

TlBr: Formula weight 284.31. Pale green-yellow; darkens in light. M.p. 456°C, b.p. 815°C. Solubility in water 2.38  $\cdot 10^{-2}$  g. (0°C); 4.76  $\cdot 10^{-2}$  g. (20°C); 20.4  $\cdot 10^{-2}$  g. (60°C) per 100 g. H<sub>2</sub>O. d 7.5. B2 structure type.

Tll: Formula weight 331.31. Two enantiotropic modifications. Yellow, rhombic (layer lattice) below ~ 168°C. d 7.29. Discolors in light. M.p. 440°C, b.p. 824°C. Solubility in water (20°C)  $6.3 \cdot 10^{-3}$  g. per g. H<sub>2</sub>O. Red, B2 structure type above ~ 168°C. d (x-ray) 7.45; supercools.

**REFERENCES:** 

- E. Cohen and K. Piepenbroek, Z. phys. Chem. (A) 167, 370 (1933).
- O. Honigschmid, L. Birkenbach and E. Kothe, Ber. Bayr. Akad. 1922, 180.
- O. Honigschmid and H. Striebel, Z. anorg. allg. Chem. <u>194</u>, 295 (1930).
- H. P. Cady and R. Taft, J. Phys. Chem. 29, 1071 (1925).

## Thallium(III) Chloride

 $TlCl_3$ ,  $TlCl_3 \cdot 4H_2O$ 

$$\begin{array}{rll} {\rm TlCl}_{\rm s} \cdot {\rm 4}\,{\rm H_2O} & {\rm TlCl} \, + \,{\rm Cl}_{\rm s} \, + \, {\rm 4}\,{\rm H_2O} \, = \, {\rm TlCl}_{\rm s} \cdot {\rm 4}\,{\rm H_2O} \\ & {\rm 239.9} & {\rm 70.9} & {\rm 72.1} & {\rm 382.8} \end{array}$$

According to Meyer, TlCl· $4H_2O$  is prepared by passing Cl<sub>2</sub> through a nearly boiling suspension of TlCl in a limited amount of water. After intermediate Tl(I)-Tl(III) chloro compounds have dissolved, the solution is evaporated to a thin syrup with continuous addition of Cl<sub>2</sub>. The water bath temperature should be

 $60-70^{\circ}$ C. The syrup immediately solidifies to a slurry of fine, white needles of TlCl·4 H<sub>2</sub>O on cooling in an ice bath, larger crystals form on slower cooling. The crystals are rapidly filtered and dried 24 hours over H<sub>2</sub>SO<sub>4</sub> and KOH(virtually no efflorescence). Cushman claims that TlCl<sub>3</sub>·4 H<sub>2</sub>O crystals can only be dried between filter papers if loss of water is to be avoided. Hecht states that he found no definite tetrahydrates.

PROPERTIES:

Colorless crystals, deliquescing in moist air. M.p. about  $43^{\circ}$ C. A saturated aqueous solution (about 86% TlCl<sub>3</sub> · 4 H<sub>2</sub>O at  $17^{\circ}$ C) is strongly acidic due to hydrolysis, and precipitates brown Tl<sub>2</sub>O<sub>3</sub> · xH<sub>2</sub>O when highly diluted. Readily soluble in alcohol and ether; 90-95% of the TlCl<sub>3</sub> is extracted from a 6N HCl solution of TlCl<sub>3</sub> by shaking with an equal volume of ether.

TICl<sub>3</sub>

 $\begin{array}{rl} \text{TlCl}_{3} \cdot 4 \text{ H}_{2}\text{O} &+ 4 \text{ SOCl}_{2} \ (4 \text{ COCl}_{2}) = \text{ TlCl}_{3} \ + \ 4 \text{ SO}_{2} \ (4 \text{ CO}_{2}) \ + \ 8 \text{ HCl} \\ 382.8 & 475.9 & (395.7) & 310.8 & 256.2 & (176.0) & 291.7 \end{array}$ 

The tetrahydrate is difficult to dehydrate without decomposition to TlCl. Hecht used  $SOCl_2$  or  $COCl_2$  for this purpose. These react with the water of hydration, giving  $SO_2$  or  $CO_2$  and HCl.

A few milliliters of  $Cl_2$  are condensed in an ordinary combustion tube by cooling and are then melted in contact with  $TlCl_3 \cdot 4H_2O$ and excess  $SOCl_2$ . A vigorous reaction starts as soon as the mixture reaches room temperature; it is initially controlled by cooling the mixture. Beautiful, white, hexagonal leaflets form at once, even at room temperature. When the reaction is complete, the lower end of the sealed tube is cooled with Dry Ice to reduce internal pressure and then the seal point is opened with a Bunsen flame. Most of the excess  $SOCl_2$  (b.p.  $75.7^{\circ}C$ ) is distilled from the reaction product in a ground joint flask on a water bath. The balance is removed by repeated evacuation. If  $TlCl_3$  is carefully worked up (i.e., warmed slowly to room temperature), it contains not more than 0.2% of TlCl.

Phosgene may be used as the dehydrating agent instead of  $SOCl_2$ . The conditions are the same except that the combustion tube must be heated to  $120^{\circ}C$ . The TlCl<sub>3</sub> obtained in this manner contains no noticeable traces of TlCl.

**PROPERTIES:** 

Begins to sinter at about  $140^{\circ}$ C. M.p.  $155^{\circ}$ C (with Cl<sub>2</sub> evolution). Very hygroscopic. Very soluble in water, alcohol and ether (see TlCl<sub>3</sub> • 4H<sub>2</sub>O).

**REFERENCES:** 

R. J. Meyer. Z. anorg. allg. Chem. 24, 335 (1900).

A. S. Cushman. Amer. Chem. J. 26, 511 (1901).

H. Hecht. Z. anorg. Chem. 254, 37 (1947).

E. H. Swift. J. Amer. Chem. Soc. 46, 2378 (1924).

## Tetrachlorothallium(III) Acid

## $H(TlCl_4) \cdot 3H_2O$

 $\begin{array}{rrrr} TlCl_{s} &+ & HCl &+ & 3 \\ 310.8 & & 36.5 & & 54.1 & & 401.3 \end{array}$ 

This compound is prepared by chlorinating an aqueous suspension of TlCl by Meyer's method, as described in the previous section. The resulting  $TlCl_3$  solution is treated with one mole of HCl per mole of  $TlCl_3$  and then concentrated on a water bath with continuous introduction of  $Cl_2$ . It is finally crystallized by evaporation in vacuum over  $H_2SO_4$  and KOH.

PROPERTIES:

Long, hairlike needles. Extremely hygroscopic; deliquesces in moist air, but stable in dry air.

REFERENCE:

R. J. Meyer. Z. anorg. allg. Chem. 24, 337 (1900).

## Thallium(I) Tetrachlorothallate(III)

Tl(TlCl<sub>4</sub>)

 $\begin{array}{rcl} \text{TlCl} &+ & \text{TlCl}_{8} = & \text{Tl}(\text{TlCl}_{4}) \\ 239.9 & & 310.8 & & 550.6 \end{array}$ 

According to Benrath,  $Tl(TlCl_4)$  is prepared from a boiling aqueous solution of more than nine moles of  $TlCl_3/liter$ . The thallium(I) chloride is dissolved in this solution; the product crystallizes in long, white needles on cooling. These cannot be washed since they decompose to a mixture of  $TlCl_3$  and yellow  $Tl_3(TlCl_8)$  on contact with water or water-miscible solvents.

REFERENCE:

A. Benrath. Z. anorg. allg. Chem. 93, 161 (1915); 136, 358 (1924).

872

# Thallium(I) Hexachlorothallate(III) Tl<sub>s</sub>(TlCl<sub>0</sub>)

 $3 \text{ TlCl} + \text{TlCl}_3 = \text{Tl}_3(\text{TlCl}_6)$ 719.6 310.8 1030.3

According to Meyer,  $Tl_3(TlCl_6)$  is prepared by saturating a not too dilute, boiling,  $HNO_3$ -acidified  $TlCl_6$  solution with freshly precipitated TlCl and subsequently filtering. The solution takes up three moles of TlCl per mole of  $TlCl_3$ . The  $Tl_3(TlCl_6)$  crystallizes on cooling as lustrous, lemon-yellow, hexagonal leaflets. The crystals can be recrystallized from 0.01 N HCl (not water) without decomposition; they are dried in vacuum over concentrated  $H_2SO_4$ .

PROPERTIES:

Slightly soluble in pure water: 0.54 g.  $(35^{\circ}C)$ ; 0.97 g.  $(55^{\circ}C)$  per 100 g. H<sub>2</sub>O. Hydrolytic decomposition is inhibited by addition of a few drops of HNO<sub>3</sub> or HCl. Converted to Tl(TlCl<sub>4</sub>) at concentrations lower than 9M TlCl<sub>3</sub>. m.p. between 400 and 500<sup>o</sup>C;d (unfused) 5.9.

**REFERENCES:** 

R. J. Meyer. Z. anorg. allg. chem. 24, 350 (1900);
A. Benrath. Z. anorg. allg. Chem. 93, 161 (1915); 136, 358 (1924).

### Potassium Hexachlorothallate(III)

 $K_3(TlCl_6) \cdot 2 H_2O$ 

According to Lyden,  $K_3(TlCl_6) \cdot 2H_2O$  is prepared by concentrating a 1:3 molar mixture (239.9 g.: 367.7 by weight) of TlCl and KClO<sub>3</sub> with excess concentrated HCl until crystallization occurs.

PROPERTIES:

Formula weight 570.45. Colorless. d 2.859. J3, structure type. Dehydrates at 150°C.

REFERENCES:

R. Lyden. Finska Kemistsamfundets Medd. <u>41</u>, 44 (1932); J. L. Hoard and L. Goldstein. J. Chem. Phys. <u>3</u>, 654 (1935).

## Potassium Pentachloroaquothallate(III)

 $K_2(TlCl_5H_2O) \cdot H_2O$ 

 $\begin{array}{rrr} TlCl_{3}+2\,KCl &+ 2\,H_{2}O &= K_{2}(TlCl_{5}H_{2}O)\cdot H_{2}O \\ 310.8 & 149.1 & 36.0 & 495.9 \end{array}$ 

According to Meyer,  $K_2(TlCl_5H_2O) \cdot H_2O$  crystallizes as monoclinic prisms on evaporation of solutions containing less than three moles of KCl per mole of TlCl<sub>3</sub>.

REFERENCE:

R. J. Meyer. Z. anorg. allg. Chem. 24, 343 (1900).

## **Cesium Nonachlorodithallate(III)**

 $Cs_3(Tl_2Cl_9)$ 

Pratt reports separation of  $Cs_3(Tl_2Cl_9)$  as a heavy, white precipitate of hexagonal prisms or plates on addition of 5-29 g. of CsCl to a solution of 40 g.  $TlCl_3$ . It can be recrystallized unchanged from the mother liquor or water.

PROPERTIES:

Stable in air. d 4.31. K7<sub>2</sub> structure type.

**REFERENCES:** 

J. H. Pratt. Z. anorg. allg. Chem. 9, 23 (1895).
J. L. Hoard and L. Goldstein. J. Chem. Phys. 3, 199 (1935).
H. M. Powell and A. F. Wells. J. Chem. Soc. (London) <u>1935</u>, 1008.

## Thallium(III) Bromide

 $TlBr_3 \cdot 4H_2O$ 

 $\begin{array}{rcl} TlBr &+ & Br_2 &= & TlBr_3 \cdot 4 H_2O \\ 284.3 & & 159.8 & 516.2 \end{array}$ 

According to Thomas,  $TlBr_3 \cdot 4H_2O$  is prepared by treating an aqueous suspension of TlBr with excess  $Br_2$  until solution is

complete and concentrating at  $30-40^{\circ}$ C (it starts to decompose at higher temperatures as well as at too great concentrations, turning yellow). The concentrate is cooled until crystallization occurs. Persistent supersaturation, which occurs readily, is most easily remedied by abrupt cooling of one point in the solution with a stream of CO<sub>2</sub> gas.

#### PROPERTIES:

Long, pale-yellowish needles. Readily soluble in water. M.p. about  $40^{\circ}$ C; however, decomposes in air at  $30^{\circ}$ C to give dark yellow Tl(TlBr<sub>4</sub>) and liberating H<sub>2</sub>O and Br<sub>2</sub>; in vacuum this occurs even at room temperature. Unstable when anhydrous.

**REFERENCE:** 

V. Thomas. Ann. Chim. Phys. [8] 11, 235 (1907).

## Thallium(I) Tetrabromothallate(III)

## Tl(TlBr<sub>4</sub>)

 $TlBr + TlBr_3 = Tl(TlBr_4)$ 284.3 444.1 728.4

According to Benrath,  $Tl(TlBr_4)$  separates as yellow needles on cooling aqueous solutions containing at least 0.3 mole of  $TlBr_3/liter$  and saturated at boiling with TlBr. The precipitate cannot be washed, since it decomposes to a mixture of  $TlBr_3$ and  $Tl_3(TlBr_6)$  on contact with water and water-miscible solvents; the  $Tl_3(TlBr_6)$  decomposes further to TlBr and  $TlBr_3$ . See also below.

**REFERENCES:** 

A. Benrath. Z. anorg. allg. Chem. 93, 161 (1915); 136, 358 (1924).

## Thallium(I) Hexabromothallate(III)

Tl<sub>3</sub>(TlBr<sub>6</sub>)

$$\begin{array}{rcl} TlBr_3 &+& 3 \ TlBr &=& Tl_3(TlBr_6) \\ 444.1 & & 852.9 & 1297.1 \end{array}$$

Benrath reports separation of red  $Tl_3(TlBr_e)$  crystals from an aqueous solution containing 0.15 mole of  $TlBr_3$ /liter and saturated
with TlBr at  $80^{\circ}$ C, when the solution is cooled from  $80^{\circ}$ C to  $5^{\circ}$ C. Below  $5^{\circ}$ C, Tl(TlBr<sub>4</sub>) crystallizes out.

If the 0.15M TIBr<sub>3</sub> solution is saturated with TlBr above  $80^{\circ}$ C, cooling to  $80^{\circ}$ C yields TlBr crystals. Further cooling yields Tl<sub>3</sub>(TlBr<sub>6</sub>) between  $80^{\circ}$ C and  $5^{\circ}$ C.

The above crystallization temperatures change with varying  $TlBr_3$  concentrations; e.g.,  $Tl_3(TlBr_6)$  separates from a 0.1M  $TlBr_3$  solution saturated with TlB only below 69°C.

The  $Tl_3(TlBr_6)$  decomposes to TlBr and  $TlBr_3$  on contact with water. See the preparation above.

REFERENCE:

A. Benrath. Z. anorg. allg. Chem. 93, 161 (1915); 136, 358 (1924).

## Rubidium Hexabromothallate(III)

 $Rb_3(TlBr_6) \cdot \frac{8}{7} H_2O$ 

3 RbBr	+ TlBr <sub>3</sub>	$= Rb_{3}(TlBr_{6})$
		( <sup>8</sup> / <sub>7</sub> H <sub>2</sub> O)
496.2	444.1	960.9

Pratt states that golden yellow crystals of  $Rb_3(TlBr_6) \cdot \theta_7 H_2O$ (he reports 1 instead of  $\theta_7 H_2O$ ) separate from a very concentrated solution of 50 g. of RbBr after addition of a solution of 1.5-24 g. of TlBr<sub>3</sub>.

PROPERTIES:

Very soluble in water, but converts to pale-yellow  $Rb(TlBr_4)$ . H<sub>2</sub>O on recrystallization. d 4.077. J3, structure type.

REFERENCE:

J. L. Hoard and L. Goldstein. J. Chem. Phys. 3, 654 (1935).

## Thallium Triiodide

 $TII \cdot I_2, TII_3$ 

 $\begin{array}{rrrr} TII \ + \ I_2 \ = \ TII \ \cdot \ I_2 \\ 331.3 \ \ 253.8 \ \ 585.2 \end{array}$ 

Berry, Lowry and Goldstein report preparation of well-developed  $TlI \cdot I_2$  crystals by refluxing equivalent quantities of TlI and  $I_2$  in

methanol, filtering and evaporating over  $CaCl_2$  in a vacuum desiccator.

**PROPERTIES:** 

Solid-state structure  $\text{TII} \cdot I_2$ ;  $\text{TII}_3$  in methanol solution. Black. Rhombic, isomorphic with  $\text{RbI} \cdot I_2$  and  $\text{CsI} \cdot I_2$ . Insoluble in water. The additional I content above TII is split off by water, ethanol, ether,  $\text{CCl}_4$  or KI solution.

**REFERENCE:** 

A. J. Berry, T. M. Lowry and R. R. Goldstein. J. Chem. Soc. (London) 1928, 1749.

## Thallium(I) Oxide

#### Tl<sub>2</sub>O

 $\begin{array}{rcl} 2 \operatorname{TlOH} &=& \operatorname{Tl}_2 \mathrm{O} &+& \operatorname{H}_2 \mathrm{O} \\ & 442.8 & & 424.8 & & 18.0 \end{array}$ 

According to Roth and Meichsner,  $Tl_2O$  is prepared by careful dehydration of TIOH in high vacuum at close to  $50^{\circ}C$ . No  $Tl_2O_3$  formation occurs at this temperature. Dehydration is almost complete after 2-3 days.

PROPERTIES:

Black, crystalline. Begins to volatilize in high vacuum at  $300^{\circ}$ C. B.p. at least  $1080^{\circ}$ C. Hygroscopic; reacts with water, forming TlOH; for solubility in water see TlOH. Solubility in absolute ethanol at room temperature: 4.4 mg. TlO<sub>2</sub>/100 ml. alcohol; heating produces thallium ethoxide. d 9.52.

**REFERENCE:** 

W. A. Roth and A. Meichsner. Z. Elektrochem. 38, 87 (1932).

# Thallium(I) Hydroxide

## TIOH

I.

 $\begin{array}{rcr} C_{2}H_{5}OTl + H_{2}O = TlOH + C_{2}H_{5}OH \\ 249.5 & 18.0 & 221.4 & 46.1 \end{array}$ 

Lamy reports that hydrolysis of thallium ethoxide precipitates TIOH. Thallium ethoxide can be rapidly prepared by the Fricke and Klein method (further developed by Freudenberg and Uthemann) according to the equation

using the device shown in Fig. 251. Ethanol is refluxed in ground



Fig. 251. Preparation of thallium ethoxide. in Fig. 251. Ethanol is refluxed in ground joint flask <sup>a</sup> while dry,  $CO_2$ -free air is passed through tube *c*, located in the middle of *b*, on top of which there is a reflux condenser, connected by a ground joint. Air flows out the condenser through a soda-lime tube to prevent back-diffusion of  $CO_2$  into the apparatus. Air and alcohol vapor diffuse through Tl turnings (made with a pencil sharpener) held in porcelain sieve *d*. Thus, TlOH and thallium ethoxide are formed. Both are readily soluble in warm alcohol and separate on cooling as a colorless, heavy, oily liquid (which probably represents a compound of  $TlOC_2H_5$  and TlOH) on the bottom of the flask ( $TlOC_2H_5$  d is 3.5).

For the hydrolytic decomposition the thallium ethoxide is cooled in an ice bath without separating excess alcohol (to prevent partial conversion of the TlOH to black  $Tl_{a}O$ ) and treated with an equal volume of

boiled and recooled distilled water. Alcohol is then evaporated in vacuum. Yellow, crystalline TIOH separates. DeForcrand states that the product still retains 4% moisture after 24 hours of standing in a desiccator on porous tile.

II.

 $\begin{array}{rrrr} Tl_2 SO_4 &+ & Ba(OH)_2 &= & BaSO_4 &+ & 2 \ TIOH \\ 504.8 & & 171.4 & & 233.4 & & 442.8 \end{array}$ 

According to Johnston, an aqueous solution of TIOH can be simply prepared by treating a hot solution of pure  $Tl_2SO_4$  with the calculated amount of dissolved Ba(OH)<sub>2</sub> and filtering off the precipitated BaSO<sub>4</sub>.

**PROPERTIES:** 

Yellow needles, readily turning dark. Saturated aqueous solutions are 1.15, 1.58 and 6.71N at 0, 19.5 and 99.2°C respectively. They attack glass, especially when hot. High solubility in alcohol. Strongly absorbs  $CO_2$  to form  $Tl_2CO_3$ . Vapor pressure about 13 mm. (46°C); 770 mm. (140°C); d 7.44.

**REFERENCES:** 

- I. A. Lamy. Ann. Chim. Phys. [3] <u>67</u>, 395 (1863); [4] <u>3</u>, 390 (1864); K. Freudenberg and G. Uthemann. Ber. dtsch. chem. Ges. <u>52</u>, 1509 (1919); A. Klein. Thesis, Technische Hochschule, Stuttgart, 1945; R. deForcrand. Compt. Rend. Hebd. Seances Acad. Sci. <u>176</u>, 874 (1923).
- II. J. Johnston. J. Phys. Chem. 62, 341 (1908).

## Thallium(II) Oxide

#### $Tl_2O_3$ , $Tl_2O_3 \cdot x H_2O$

2 TINO <sub>3</sub>	+ 6 KOH	$+ 2 \operatorname{Cl}_2$	$= Tl_2O_3 +$	- 2 KNO <sub>3</sub>	+ 4  KCl	$+ 3 H_2O$
532.3	336,6	141.8	456.8	202.2	298.2	54.1

According to Huttig and Mytizek, a solution of 30 g. TlNO<sub>3</sub> in one liter of water is gradually treated with 400 ml. of pure,  $CO_2$ free, dilute KOH and is then ice cooled and saturated with  $Cl_2$ . When precipitation of  $Tl_2O_3 \cdot xH_2O$  ceases, the solids are washed with a total of 40 liters of water by decantation (the hydrated oxide settles out very slowly). The precipitate is then filtered and washed until free of Cl<sup>-</sup> ions. The brown, hydrated oxide still has the composition  $Tl_2O_3 \cdot 1.47H_2O$  after 10 days of drying in an ordinary desiccator over 26%  $H_2SO_4$ . Essentially anhydrous, brown  $Tl_2O_3$  is obtained after four days of drying in vacuum over  $32\% H_2SO_4$ . Isobaric dehydration of hydrated oxide by heating even below  $90^{\circ}C$  at 10 mm. results in a small but definite loss of oxygen; this increases above  $100^{\circ}C$  to the point where the completely dehydrated product, obtained by heating at  $330^{\circ}C$  has only the composition  $Tl_2O_{2\cdot8}$ . Some elemental Tl also begins to sublime at  $330^{\circ}C$ .

#### **PROPERTIES:**

 $Tl_2O_3$ : brown to black, depending on preparative conditions. M.p. 717°C under 1 atm. of  $O_2$  Insoluble in water.d (x-ray) 10.11. C sesquioxide structure type.

#### **REFERENCES:**

 G. F. Huttig and R. Mytizek, Z. anorg. allg. Chem. <u>192</u>, 187 (1930); compare also W. O. Milligan and H. B. Weiser. J. Amer. Chem. Soc. <u>59</u>, 1673 (1937), and H. B. Weiser and W. O. Milligan. J. Phys. Chem. <u>42</u>, 673 (1938).

## Thallium Sulfides

## THALLIUM (I) SULFIDE Tl<sub>2</sub>S

I.

 $2 \text{ Tl} + S = \text{Tl}_2 S$ 408.8 32.1 440.8

According to Ketelaar and Gorter,  $Tl_2S$  is formed by fusing equimolar quantities of Tl and S in a sealed,  $H_2$ -filled tube at room temperature and 60 mm. A brown-black button with a metallic luster is formed.

II.  $2 \operatorname{TlOC}_2H_5 + H_2S = \operatorname{Tl}_2S + 2 \operatorname{C}_2H_5OH$ 498.9 34.1 440.8 92.1

According to Reuter and Goebel, finely divided, very reactive, oxygen-free  $Tl_2S$  can be prepared by reaction of dry  $H_2S$  with thallium ethoxide dissolved in absolute alcohol.

The ethoxide obtained from 10 g. of TI [see thallium(I) hydroxide, p. 877, for preparation] is dissolved in 250 ml. of absolute alcohol and passed through a fluted filter into the flask of a distillation apparatus. Air is displaced with pure N<sub>2</sub>, followed by dry H<sub>2</sub>S, resulting in a finely divided Tl<sub>2</sub>S precipitate. Air is prevented from entering the apparatus by having the condenser tube reach almost to the bottom of the receiver, which contains some alcohol.

The reaction is over within two hours. The gas inlet tube is closed, the alcohol distilled off on a water bath, and the  $Tl_2S$  dried by heating on a water bath for two hours and finally in oil pump vacuum.

#### **PROPERTIES:**

M.p.  $448.5^{\circ}$ C under N<sub>2</sub>. Volatilization begins above  $300^{\circ}$ C. Practically insoluble in water. d (x-ray) 8.39. PbI<sub>2</sub> structure (C6 type) with only small deviations. Finely powdered, deep black Tl<sub>2</sub>S smears like graphite when rubbed.

## THALLIUM (I, III) SULFIDE Tl<sub>2</sub>S • Tl<sub>2</sub>S<sub>3</sub> or TlS

$$Tl + S = TlS$$
  
204.4 32.1 236.5

Hahn and Klingler prepared "TIS" by reaction of equimolar quantities of pure Tl with redistilled S in an evacuated Pyrex tube heated to Bunsen burner temperature. The product is kept 24 hours at  $200^{\circ}$ C for better crystallization.

## PROPERTIES:

Black to steel-gray with metallic luster. d (x-ray) 7.61. B37 structure type.

**REFERENCES:** 

- J. A. A. Ketelaar and E. W. Gorter. Z. Kristallogr. <u>101</u>, 367 (1939);
- B. Reuter and A. Goebel. Z. anorg. allg. Chem. 268, 101 (1952);
- H. Hahn and W. Klingler. Z. anorg. Chem. 260, 110 (1949).

## Thallium(I, III) Selenide

## $Tl_2Se \cdot Tl_2Se_3$ , TISe

Tl + Se = TlSe204.4 79.0 283.4

According to Ketelaar, Hart, Moerel and Polder,  $Tl_2Se_3$ .  $Tl_2Se_3$  or "TlSe" is prepared by fusing equimolar quantities of Tl and S at 400°C and cooling very slowly, a procedure similar to preparation of  $Tl_2S$ . The button fractures under light pressure into tetragonal prisms with a very strong metallic luster.

**PROPERTIES:** 

M.p.  $310^{\circ}$ C (congruent). Indefinitely stable in dry air. d (x-ray) 8.31. B37 structure type.

**REFERENCE:** 

J. A. A. Ketelaar, W. H. T. Hart, M. Moerel and D. Polder. Z. Kristallogr. <u>101</u>, 396 (1939).

#### Thallium(I) Sulfate

Tl<sub>2</sub>SO<sub>4</sub>

This compound is prepared by dissolving pure Tl in warm, moderately dilute  $H_2SO_4$  and concentrating until crystallization begins.

Since Tl dissolves more easily in hot, dilute  $HNO_3$ , it is advantageous to prepare the sulfate by evaporating a nitric acid solution containing a small excess of concentrated  $H_2SO_4$ . The small amounts of Tl(III) sulfates formed are reduced with  $SO_2$  (see also Purification of Technical Grade Thallium, p. 869).

Ensslin reports (private communication) formation of Tl(I) sulfate by passing a mixture of steam and air countercurrently to sulfuric acid (150 g. of  $H_2SO_4$ /liter) through a column filled with crude, granulated thallium. The hot effluent, which is almost saturated with  $Tl_2SO_4$ , is filtered on a heated suction filter into a crystallization flask. The mother liquor is reacidified with  $H_2SO_4$ and recycled through the column. In this manner, pure thallium can be obtained via Tl(I) sulfate, starting with crude thallium containing 8% Pb and 1% Cu.

PROPERTIES:

M.p.  $632^{\circ}C$ ; volatilizes undecomposed. Solubility in water 2.7 (0.2°C); 4.3 g. (15.6°C); 18.5 g. (100°C) per 100 g. of H<sub>2</sub>O. d 6.765. Isomorphous with K<sub>2</sub>SO<sub>4</sub> (H 16 structure type).

## Disulfatothallic(III) Acid

## $HTl(SO_4)_2 \cdot 4 H_2O$

 $\begin{array}{rl} Tl_2O_8 \ + \ 4 \ H_2SO_4 \ + \ 5 \ H_2O \ = \ 2 \ HTl(SO_4)_2 \ \cdot \ 4 \ H_2O \\ 456.8 \ & 392.3 \ & 90.1 \ & 939.2 \end{array}$ 

Meyer and Wilk report preparation of  $HTl(SO_4)_2 \cdot 4H_2O$  by saturating warm 45-50%  $H_2SO_4$  with  $Tl_2O_3$  or, better still, with  $Tl_2O_3 \cdot xH_2O$  (because of its higher rate of solution). Concentration of the clear solution on a water bath yields heavy, colorless crystals, which are dried several days on heated porous tile in a steam-heated drying oven. The crystals cannot be washed with water due to their tendency to hydrolyze. Temperature and concentration of adhering mother liquor may not be changed during drying, to avoid decomposition set-in.

Formula weight 469.58.

REFERENCE:

J. Meyer and H. Wilk. Z. anorg. allg. Chem. 132, 244 (1924).

# Thallium(III) Hydroxide Sulfate

## $Tl(OH)SO_4 \cdot 2H_2O$

 $\begin{array}{rrr} Tl_2O_3 \ + \ 2H_2SO_4 \ + \ 3\,H_2O \ = \ 2\,Tl(OH)SO_4 \cdot 2\,H_2O \\ 456.8 \ 196.2 \ 54.1 \ 707.0 \end{array}$ 

According to Meyer and Wilk,  $Tl(OH)SO_4 \cdot 2H_2O$  is formed analogously to  $HTl(SO_4)_2 \cdot 4H_2O$ , by using 20%  $H_2SO_4$  instead of 45-50% acid, or when  $HTl(SO_4)_2 \cdot 4H_2O$  is crystallized from 20%  $H_2SO_4$ . Drying is carried out in the same way as in the previous preparation to avoid hydrolytic decomposition.

Formula weight 353.49.

**REFERENCE:** 

J. Meyer and H. Wilk. Z. anorg. allg. Chem. 132, 244 (1924).

## Thallium(I) Nitride

Tl<sub>3</sub>N

Franklin reports separation of  $Tl_3N$  as a black precipitate if a solution of  $TlNO_3$  in liquid  $NH_3$  is treated with the corresponding quantity of  $KNH_2$  in liquid  $NH_3$ .

**PROPERTIES:** 

Explodes on contact with water or dilute acids, also on heating or impact. Soluble in solutions of  $TINO_3$  or  $KNH_2$  in liquid  $NH_3$ . The sample for analysis is washed with liquid  $NH_3$ , vacuum dried by evaporation of  $NH_3$  and carefully hydrolyzed with steam.

**REFERENCE**:

E. C. Franklin, J. Phys. Chem. 16, 683 (1912).

#### Thallium(I) Nitrate

#### TlNO<sub>3</sub>

 $Tl + HNO_3 = TlNO_3 + \frac{1}{2}H_2$ 204.4 63.0 266.4 1.0

Pure Tl is dissolved in dilute  $HNO_3$ , excess acid evaporated at  $110^{\circ}C$ , and the nitrate repeatedly crystallized.

**PROPERTIES:** 

M.p. 206°C, b.p. 433°C; d 5.55. Three enantiotropic modifications: rhombic below about 75°C, hexagonal between about  $75^{\circ}$ C and about  $142^{\circ}$ C, and cubic above about  $142^{\circ}$ C. Solubility in water at 0, 15, 30 and 104.5°C (b.p.): 4, 8, 14 and 593.9 g. of TlNO<sub>3</sub>/100 g. of H<sub>2</sub>O. Densities of these solutions at the same temperatures: 1.035, 1.065, 1.115 and 3.191.

**REFERENCES:** 

C. van Eyk. Z. phys. Chem. <u>30</u>, 430 (1899). Earl of Berkeley. Phil. Trans. (A) <u>203</u>, 213 (1904).

## Thallium(I) Carbonate

Tl<sub>2</sub>CO<sub>3</sub>

2 TlOH	+	$\rm CO_2$	=	$Tl_2CO_3$	+	$H_2O$
442.8		44.0		468.8		18.0

This compound is prepared by saturating a hot aqueous solution of TIOH with  $CO_2$  and evaporating on a water bath. The  $Tl_2CO_3$ , which separates on cooling, is filtered by suction on a Pyrex filter, recrystallized from hot water (it is best to work in alkali-stable glass vessels) and dried at  $105^{\circ}C$ .

PROPERTIES:

Colorless, needle-shaped, monoclinic crystals with sharp enantiotropic transition point at 228°C. M.p. 272°C. Solubility in water 5.2 g. (18°C); 22.4 g. (100.8°C) per 100 g. of  $H_2O$ ; aqueous solutions are strongly basic due to hydrolysis. Insoluble in absolute alcohol. Not hygroscopic, stable in air up to 175°C without noticeable change. Dissociation pressure of the melt 1 atm. at 368°C. d 7.16.

**REFERENCES:** 

K. Freudenberg and G. Uthemann. Ber. dtsch. chem. Ges. <u>52</u>, 1509 (1919).

# Thallium(II) Formate, Thallium(I) Malonate, Clerici's Solution

Clerici's solution is best prepared starting with the two salts or their solutions separately.

884

#### A) THALLIUM (I) FORMATE

$$Tl_{2}CO_{3} + 2 HCOOH = 2 HCOOTl + H_{2}O + CO_{2}$$

$$468.8 \quad 92.1 \quad 498.8 \quad 18.0 \quad 44.0$$

$$TlOH + HCOOH = HCOOTl + H_{2}O$$

$$221.4 \quad 46.0 \quad 249.4 \quad 18.0$$

According to Vhay and Williamson, thallium(I) formate is prepared by treating  $Tl_2CO_3$  with formic acid. Excess formic acid is removed by repeated evaporation to dryness on a water bath and redissolving in distilled water until the odor of formic acid disappears.

According to Brauer and Haag, an aqueous solution of TIOH (see p. 877) is prepared, treated with sufficient formic acid to render the solution neutral or weakly acid, and evaporated in vacuum. Evaporation in open porcelain dishes is not recommended, since the residue darkens and becomes partly insoluble.

Brauer and Haag prefer the following procedure as more reliable, even though it is somewhat more tedious: about 30 g. of Tl is cut in small pieces and heated with 150 ml. of about 90% formic acid in a flask with a ground joint and a 1-m.-long air condenser. The heating must be just sufficient to reflux the formic acid gently. Every 12 hours, 10-20 ml. of formic acid is added. Practically all the Tl reacts after about 60 hours and the solution can be evaporated in vacuum.

#### PROPERTIES:

M.p.  $101^{\circ}$ C (without decomposition). Melt colorless and mobile. d ( $104^{\circ}$ C) 4.967. Hygroscopic, very soluble in water (500 g. of TlOOCH/100 g. of H<sub>2</sub>O at  $10^{\circ}$ C), forming highly mobile, colorless, odorless, completely stable liquids with high densities. Densities of saturated solutions at 10, 20, 50 and  $90^{\circ}$ C: 3.31, 3.40, 4.10 and 4.76. Soluble with difficulty in alcohol. Readily soluble in methanol, from which it can be purified by crystallization.

#### B) THALLIUM (I) MALONATE

Thallium malonate can be prepared by treating  $Tl_2CO_3$  with an equimolar quantity of malonic acid dissolved in water (solubility: 73.5% at 20°C) and evaporating. Thallium malonate deliquesces in air.

## C) CLERICI'S SOLUTION

Clerici's solution is prepared by dissolving equal weights of thallium formate and malonate in the minimum amount of water.

The density of the saturated solution at  $20^{\circ}$ C is 4.324. The solution density may be reduced by addition of water and increased by evaporation. The refractive index is a strictly linear function of the density at constant temperature.

Clerici's solution which has picked up impurities in use and thus become brown (due to malonic acid decomposition products) can be regenerated, according to Rankama, by dilution of the concentrated solution to four times its volume with distilled water, treating with 1 g. of powdered charcoal per 100 g. of concentrated solution, heating, stirring and filtering. The solution is then clear. Thallium is precipitated from it as TICl and converted to  $Tl_2SO_4$ with  $H_2SO_4$ . The latter is then converted to TlOH (see p. 877) by treatment with Ba(OH)<sub>3</sub>, which is in turn converted to  $Tl_2CO_3$  (see p. 884) by saturation of the hot solution with  $CO_2$  and evaporation. After recrystallization, the  $Tl_2CO_3$  is used as described above for the preparation of thallium formate and malonate.

**REFERENCES:** 

- E. Clerici. Atti R. Accad. d. Lincei, Roma [5] <u>16</u>, 1, 187 (1907); see also Z. Krystallogr. 46, 392 (1909);
- J. S. Vhay and A. T. Williamson. Amer. Mineralogist <u>17</u>, 561 (1932);
- G. Brauer and H. Haag. Private communication (1952);
- R. Jahns. Amer. Mineralogist 24, 116 (1939);
- C. J. Payne, L. Franklin and B. W. Anderson. Gemmologist <u>5</u>, 274 (1936);
- K. Rankama. Bull. Commiss. Geol. Finlande 9, 65 (1936).

# SECTION 17

# Alkaline Earth Metals

## P. EHRLICH

## Beryllium

Be

On a laboratory scale, beryllium is prepared by electrolysis of a mixture of molten beryllium fluoride and alkali or alkaline earth fluorides. The product obtained is 99.7% pure; the commercial material obtained by the same method is < 99%. Industrial electrolysis of beryllium chloride-alkali chloride melts yields beryllium with a purity usually greater than 99.8%.

Very high purity Be is obtained by Kroll's procedure (which is a modification of Sloman's method) which consists in vacuum distillation in the apparatus shown in Fig. 252. The relatively impure beryllium is placed in a BeO crucible and induction heated, and its vapor is condensed in a condensation hood. The latter is tightly joined to the crucible, but has an opening on top, so that the course of the distillation may be observed through a quartz peephole at the top of the apparatus. The BeO crucible is embedded in sintered, powdered BeO, which fills a retaining vessel of sintered alumina.

The whole assembly is placed in a quartz tube which is closed off on top by a special water-cooled adapter. This metallic adapter may be made either of compressed metal powder or of bronze impregnated with tin under vacuum. It is sealed to the quartz tube with high-vacuum silicone grease. The apparatus is connected to a high-vacuum pump by means of a flexible tombac tube. The distillation requires a vacuum of  $< 10^{-3}$  mm.

The distillation of crude beryllium is connected with certain difficulties since the metal occludes the electrolyte and alkaline earth metals. These may cause bubbling and spattering of the metallic melt during the distillation. Therefore, preliminary degassing is carried out without the condensation hood until gas evolution ceases. After cooling, the condensation hood is installed, and after high vacuum has been restored the distillation is carried out at a temperature of 1400 to 1500°C. If beryllium flakes are used, they are first compressed and melted. This melt may be



Fig. 252. High-vacuum distillation of beryllium.

distilled at once since it does not contain salt occlusions. The distillation is complete in about 20 minutes. The run is ended when about 75% of the charge has distilled. Depending on the operating temperature, the beryllium collects on the hood in the form of large globules or an incrustation (dendrites). The degree of purity should exceed 99.97%, neglecting the small quantity of oxygen that may be present.

The recovery of Be compounds from gadolinite is described in the section on scandium, yttrium and rare earth metals.

**PROPERTIES:** 

Atomic weight 9.02.

M.p.  $1280^{\circ}$ C, b.p. 2970°C; crystallizes in structure type A3; brittle at room temperature, ductile at red heat.

In contact with water, Be becomes covered with a thin oxide layer but is not attacked further; however, it dissolves very vigorously in dilute acids.

**REFERENCES:** 

- H. Funk, Die Darstellung der Metalle im Laboratorium [The Preparation of Metals in the Laboratory], Stuttgart, 1938, p. 28.
- W. Kroll, Beryllium, in: A. E. van Arkel, Reine Metalle [Pure Metals], Berlin, 1939, p. 99.
- W. Kroll, Metallwirtschaft 13, 725 (1934); Metal. Ind. <u>47</u>, 29 (1935).

888

H. A. Sloman, J. Inst. Met. 44, 365 (1932).

# Beryllium Chloride BeCl<sub>2</sub>

I.

Beryllium halides do not attack Pyrex glass if the temperature is carefully controlled. Therefore, only the actual reaction apparatus must be of quartz, while vacuum sublimation and bottling may be carried out in Pyrex glass.

As shown in Fig. 253, the apparatus consists of a large quartz tube A (25-mm. diameter, 400 mm. long), which is connected with a gas drying system by means of a large ground glass joint; its other end is connected, by means of a small ground glass joint a, with a Pyrex apparatus. The latter consists of several cylindrical chambers B, C, D and E which serve as receivers and resublimation vessels for the halide. Chamber D is fitted with a side tube closed off by a breakable bulb. A system of storage bulbs (1, 2, 3, 4) is later sealed to the side arm.



of beryllium chloride.

An intimate blend of 3.5 g. of BeO and 2.5 g. of charcoal made from calcined sugar is charged into a quartz boat s, which is then pushed into tube A. After removal of water and other impurities adsorbed on the carbon by heating to about 900°C in a stream of N<sub>2</sub>, the nitrogen is replaced by  $Cl_2$  and the reaction temperature is adjusted to 700°C. The sublimate of BeCl<sub>2</sub>, which collects over a period of several hours, forms beautiful white, matted crystals in the rear section of tube A. After completion of the reaction, pure N<sub>2</sub> is reintroduced and the Pyrex apparatus is connected to joint a. The latter is not greased since no vacuum is applied. By proper positioning of the tubular furnace, the halide resublimes and deposits in the first chamber *B*, while leaving a residue in A. The sublimation takes place at an adequate rate at  $380^{\circ}$ C. Joint a is then melt-sealed at its narrowest point and the apparatus is connected to the high-vacuum pump. A forerun is then sublimed into the fourth chamber *E* by shifting the furnace. The furnace is then shifted back and the bulk of the material is sublimed into chamber C. This may be done at 330°C, provided a high vacuum is maintained. Chamber B, still containing a residue, is removed by sealing off.

The material is then resublimed from C into D. Thus, only chamber D, completely evacuated and sealed off on both sides, remains of the entire Pyrex apparatus. The vacuum pump must operate continuously during these sublimations since it has been found impossible to carry them out in an evacuated and sealed apparatus. Small amounts of gas are desorbed from the glass walls, thus decreasing the vacuum. This in turn raises the sublimation temperature. At these higher temperatures, traces of BeCl<sub>2</sub> react with the glass, forming BeO + SiCl<sub>4</sub>, which again leads to deterioration of the vacuum until the sublimation ceases altogether and liquid SiCl<sub>4</sub> forms at the coldest spot. It is best to carry out the entire operation without interruption over a period of about 30 hours.

To bottle  $\operatorname{BeCl}_2$ , a 10-cm.-long glass tube is attached to the side tube of D, a small glass-covered steel rod b is pushed in, and finally the four-bulb assembly is sealed on. The apparatus is connected to the vacuum pump via a ground-glass joint, a cold trap and a stopcock. After thorough heating, the stopcock is closed, the apparatus is detached from the pump, and the little glass bulb is broken by moving b with a magnet. The beryllium chloride is sublimed into the individual bulbs at the highest vacuum obtainable, while the trap is cooled with liquid nitrogen; this operation requires about four hours per bulb.

Other preparative methods: The reaction of beryllium with  $Cl_2$  and HCl (II) or the treatment of BeO with  $CCl_4$  (III) may also be recommended.

#### PROPERTIES:

Snow-white crystals or crystalline mass. M.p.  $405^{\circ}$ C, b.p.  $488^{\circ}$ C; d 1.90. Extremely hygroscopic; dissolution in water is highly exothermic. The solution is strongly acid due to hydrolytic cleavage. On evaporation, the hydrate BeCl<sub>2</sub> · 4H<sub>2</sub>O crystallizes in monoclinic, deliquescent platelets if the hydrolysis is depressed by the addition of hydrochloric acid; otherwise, basic chlorides precipitate. Anhydrous BeCl<sub>2</sub> is readily soluble in alcohol and ether.

#### **REFERENCES:**

I. O. Hönigschmid and T. Johannsen, Z. Naturforsch. <u>1</u>, 650 (1946); H. Remy, Lehrbuch d. anorg. Chem. [Textbook of Inorganic Chemistry] vol. I, Leipzig 1950, p. 257.

- II. O. Rahlfs and W. Fischer, Z. anorg. allg. Chem. 211, 349 (1933).
- III. J. Besson, Comptes Rendus Hebd. Séances Acad. Sci. <u>214</u>, 861 (1942).

## **Beryllium Bromide**

## BeBr<sub>2</sub>

I.

 $\begin{array}{rrrr} BeO + C + Br_2 = BeBr_2 + CO \\ 25.0 & 12.0 & 159.8 & 168.8 & 28.0 \end{array}$ 

The directions for the preparation of BeCl<sub>2</sub> apply with the following modifications: a nitrogen stream charged with Br<sub>2</sub>vapor is passed over the BeO + C mixture at 1200°C. This mixture is very corrosive to quartz. The first sublimation may be completed sufficiently rapidly at atmospheric pressure at  $360^{\circ}$ C; for the following sublimations a temperature of  $310^{\circ}$ C is sufficient, provided a high vacuum is maintained.

II.  $B_{e} + B_{r_{2}} = B_{e}B_{r_{2}}$ 9.0 159.8 168.8

To prepare  $\operatorname{BeBr}_2$  directly from the elements, bromine is evaporated in a flask by heating with an infrared lamp and passed in a stream of argon over Bepowder in a quartz tube at a temperature of 550°C. The end of the tube protrudes into a cold receiver consisting of a large glass cylinder. In order to avoid plugging of the tube in the transition zone, the interior of the tube is heated with an electrical winding to 450–500°C; the winding is surrounded by a small quartz tube which is sealed at one end. This quartz tube extends from the receiver into the hot zone of the reaction tube. The receiver is separated from the atmosphere by several wash bottles, which also serve as bubble counters.

The anhydrous bromide product is transferred to storage vessels under an inert gas blanket. The yield is 200-250 g. in six hours.

#### **PROPERTIES:**

Long, white needles. M.p.  $488^{\circ}$ C, with sublimation beginning at  $360^{\circ}$ C; d 3.47. Very hygroscopic; large heat of solution. The tetrahydrate crystallizes as rod-shaped, hygroscopic crystals when the BeBr<sub>2</sub> solution is evaporated to a sirupy consistency. Gaseous HBr must be added to the solution to avoid precipitation of basic salts due to hydrolysis.

REFERENCES:

- I. O. Hönigschmid and T. Johannsen, Z. Naturforsch. 1, 650 (1946).
- II. G. B. Wood and A. Brenner, J. Electrochem. Soc. 104, 29 (1957); see also O. Rahlfs and W. Fischer, Z. anorg. allg. Chem. 211, 349 (1933).

# Beryllium Iodide

## BeI<sub>2</sub>

I.

Be	$+ I_2 =$	$\operatorname{Bel}_2$
9.0	253.8	262.8

Beryllium iodide is formed via the reaction of gaseous  $I_2$  (or a mixture of pure  $H_2$  and  $I_2$ ) with metallic Be in a quartz tube at dull red heat. It is best to work in a sealed tube; however, the metal and the solid iodine should not be blended with each other, since this may cause bursting of the tube during heating. The metal is inserted separately in a glass container so that it comes in contact only with the  $I_2$  vapor. After 2-3 days at 480°C, little unreacted iodine remains. Thus, each tube contains 4 g. of Be (1.8 g. excess) and 63 g. of iodine. The subsequent sublimation, which gives a pure white product, is carried out as described for BeCl<sub>2</sub>; however, it is preferable to work in all-quartz equipment in view of the higher temperatures.

п.

Be <sub>2</sub> C -	$+ 2 I_2 = 507.7$	2 Bel <sub>2</sub>	+ C
30.0		525.7	12.0
${}^{{ m Be_2C}}_{30.0}$ +	$\begin{array}{l} 4 \hspace{0.5mm} \mathrm{HI} \hspace{0.5mm} = \\ 511.7 \end{array}$	2 BeI <sub>2</sub> 525.7	+ CH <sub>4</sub> 16.0

The iodide cannot be prepared in the same fashion as the chloride, i.e., by passing a nitrogen stream charged with  $I_2$  over a mixture of BeO + C; Be<sub>2</sub>C is required as the starting material.

In a quartz tube at about 700°C,  $Be_2C$  is allowed to react with carefully purified and dried HI or with a hydrogen stream containing  $I_2$  vapor. The subsequent purification by sublimation is the same as that described above for  $BeCl_2$ .

**PROPERTIES:** 

White needles. M.p. 480°C, b.p. 488°C. Very hygroscopic; hydrolyzes vigorously, evolving HI.

**REFERENCES:** 

- I. G. B. Wood and A. Brenner, J. Electrochem. Soc. <u>104</u>, 29 (1957); see also O. Rahlfs and W. Fischer, Z. anorg. allg. Chem. 211, 349 (1933).
- II. P. Lebeau, Ann. Chim. [7] <u>16</u>, 476, 490 (1899); C. Messerknecht and W. Biltz, Z. anorg. allg. Chem. 148, 152 (1925).

## Beryllium Oxide and Beryllium Carbonate

#### BeO, BeCO<sub>3</sub>

 $\begin{array}{rcl} \text{BeCO}_3 &= & \text{BeO} &+ & \text{CO}_2 \\ 69.0 & & 25.0 & & 44.0 \end{array}$ 

Commercial beryllium carbonate, which usually contains more than 1% of impurities, is mixed in a platinum vessel with hot, twice-distilled acetic acid and the hot solution is filtered through a platinum Gooch crucible. The basic acetate separates as fine crystals on rapid chilling; it is recrystallized from glacial acetic acid three times. Each time, the salt is separated by means of a platinum basket centrifuge. After this preliminary purification, the acetate is sublimed at a temperature of  $250^{\circ}$ C in a largediameter glass tube with a stream of pure, dry air flowing through the apparatus. The sublimate is heated in a degassed Pyrex flask with pure concentrated nitric acid to convert it to the nitrate. The latter may be calcined to the oxide at  $1100^{\circ}$ C; however, the oxide thus prepared always contains about 0.35 ml. of gas (N<sub>2</sub>, O<sub>2</sub>) per gram of substance.

Very pure oxide may be obtained when the beryllium nitrate prepared as described above is reconverted to the carbonate. An acid solution of the nitrate is evaporated in a platinum dish to remove excess acid, the residue is dissolved in some water, and enough distilled ammonium carbonate solution is added to redissolve the initially precipitated beryllium carbonate and give a clear solution. The solution is then evaporated in a platinum dish until all the beryllium carbonate has separated as a coarse precipitate. After thorough washing with pure water and then with freshly distilled alcohol, the product is dried and finally calcined to the oxide in a platinum vessel placed in an electric furnace at 900°C.

#### PROPERTIES:

White, loose powder. M.p. 2530°C; d 29. Very sparingly soluble in water (about 0.20 g. per liter). Its solubility in acids depends upon the calcination temperature; dissolves most readily in hydrofluoric acid. Crystallizes in structure type B4. REFERENCE:

O. Hönigschmid and T. Johannsen, Z. Naturforsch. 1, 650 (1946).

## Beryllium Hydroxide

## Be(OH)<sub>2</sub>

Two crystalline modifications may be obtained, the metastable  $\alpha$ -Be(OH)<sub>2</sub> and the stable  $\beta$ -Be(OH)<sub>2</sub>.

 $\alpha$ -Be(OH)<sub>2</sub>

The  $\alpha$  form is the primary product of aging of amorphous beryllium hydroxide, which is obtained by the precipitation of a beryllium salt solution with ammonia in the absence of CO<sub>2</sub>;  $\alpha$ -Be(OH)<sub>2</sub> is then obtained by prolonged heating (about 24 hours) of the amorphous precipitate with 10% ammonium hydroxide solution.

A very pure material may be obtained by electrodialysis of amorphous beryllium hydroxide. This is a slow procedure and takes several days.

 $\beta$ -Be(OH)<sub>2</sub>

The beta modification is best prepared by slow hydrolysis of sodium beryllate. Boiling 10N NaOH solution is saturated with pure, amorphous beryllium hydroxide until a permanent turbidity is just evident. A sandy, finely crystalline product separates upon slow cooling. Under the microscope, the crystals appear as beautiful, regular double pyramids. They may be purified, without changing their appearance, with warm water until they no longer show an alkaline reaction; the crystals are then dried at  $80^{\circ}C$ .

The same mother liquor may be reused several times in the saturation and crystallization sequence.

#### **PROPERTIES:**

Formula weight 43.04. d 1.92. Crystalline beryllium hydroxide is very slightly soluble in water and in dilute alkali. Both forms are readily soluble in hot concentrated sodium hydroxide.

#### REFERENCES:

L. Havestad and R. Fricke, Z. anorg. allg. Chem. 188, 357 (1930).

- R. Fricke and B. Wullhorst, Z. anorg. allg. Chem. 205, 127 (1932).
- R. Fricke and G.F. Huttig, Hydroxyde und Oxydhydrate [Hydroxides and Hydrated Oxides], Leipzig 1937, p. 12, as well as the literature cited there.

## Sodium Beryllates

I. Concentrated sodium hydroxide saturated with beryllium hydroxide, or alcoholic potassium hydroxide saturated with potassium beryllate, both prepared with exclusion of  $CO_2$ , is filtered in the presence of KOH through an asbestos filter in a silver funnel. The filtrate is vacuum-evaporated in a nickel dish in the presence of  $H_2SO_4$  and KOH. The first precipitate consists of  $Na_2CO_3$  and some Be(QH)<sub>2</sub>. As soon as the separation of the snow-white, shining sodium beryllate begins, the filtration is repeated and the solution further evaporated. The product is washed with alcohol and dried in a vacuum desiccator.

II. Monosodium beryllate and  $Be(OH)_2$  exist as the solid-phase components in the system BeO-NaOH-H<sub>2</sub>O at 30°C, when the concentrations of NaOH and BeOH are about 33% and 4.3%, respectively; at higher sodium hydroxide concentrations, monosodium beryllate is the only solid-phase component.

III. Sodium orthoberyllate with the formula  $Na_4BeO_3$  is obtained as a white powder via the reaction of  $Na_2O$  with BeO in a silver boat at 500°C. For techniques of working with exclusion of  $CO_2$  and atmospheric moisture, see Part I, the section on orthostannates and the original literature.

PROPERTIES:

Strongly hygroscopic crystals, decomposed by atmospheric  $CO_2$ .

REFERENCES:

- I. G. Krüss and H. Moraht, Liebigs Ann. Chem. 260, 174 (1890).
- II. R. Fricke, A. Münchmeyer and F. Engelhardt, Z. anorg. allg. Chem. <u>166</u>, 247 (1927).
- III. E. Zintl and W. Morawietz, Z. anorg. allg. Chem. 236, 372 (1938).

## **Beryllium Sulfide**

#### BeS

Beryllium sulfide may be prepared either by (I) synthesis from the elements, (II) reaction of the chloride with  $H_2S$  (which, however, does not yield chlorine-free material), or (III) from BeSO<sub>4</sub> or BeO.

The last two methods should be used only when metallic Be is unavailable.

I.

$$Be + S = BeS$$
  
9.0 32.1 41.1

Sulfur vapor mixed with  $H_2$  is passed over pure Be at a temperature of about 1150°C. The ground glass apparatus, shown in Fig.

254, consists essentially of a quartz tube with an enlargement at one end to serve as a sulfur receiver; the Be is contained in a boat made of quartz or, still better, of  $Al_2O_3$  or BeO, and placed at the center of the quartz tube. To obtain a quantitative reaction, the product of the first run is finely pulverized in an agate mortar and again reacted with the sulfur vapor.



Fig. 254. Preparation of beryllium sulfide.

Von Wartenberg simplified the process as follows: a quartz tube of 15-mm, diameter and 700-mm, length, sealed at the lower end, is charged with 6 g. of Be and 30 g. of distilled S; the tube is slipped into a platinum-wound furnace, which is inclined at a 45° angle; the Be is first ground in a coffee mill and screened to give particles 0.2-0.5 mm. in diameter. The furnace is heated to 1350°C for about two hours; the sulfur evaporates constantly and flows back down the protruding portion of the tube, which acts as a reflux condenser. The sulfur vapor thus excludes air. After removal and cooling of the tube, its lower end is cut off and the easily removed, sintered cake is crushed and again treated with sulfur in the same fashion. The cut-off tube end may be resealed on a new tube section. After the cake is freed of sulfur under vacuum, it still contains Be flakes, which, however, can be separated by pulverizing and screening (0.1-mm. screen) to such an extent that the dirty yellow powder, which still retains a faint odor, no longer evolves H<sub>2</sub> when added to dilute acids. Analysis shows a BeS content of 98% on the basis of the determination of  $H_2S$  evolved; however, 2.4% of the product is insoluble in dilute H2SO4.

II. 
$$BeCl_2 + H_2S = BeS + 2 HCl$$
  
79.9 34.1 41.1 72.9

Beryllium chloride, obtained by passing HCl over a red-hot mixture of BeO + C, is purified as thoroughly as possible by repeated sublimation in a quartz tube. Without opening the apparatus, the BeCl<sub>2</sub> is reacted with pure  $H_2S$  (prepared from S and  $H_2$ ). Too rapid sublimation of the BeCl<sub>2</sub> (at about 400°C) must be avoided by

increasing the temperature only very gradually if a satisfactory yield is to be obtained. Furthermore, the reaction must be localized by means of a water-cooled quartz tube, which is inserted in the quartz reaction tube. The chloride should be sublimed in the  $H_2S$  stream back and forth several times. The product, which still has a high chlorine content, is heated at 850 to 950° in the  $H_2S$  stream for an additional hour. This yields a grayish, amorphous sulfide which still contains traces of chloride and which is less stable in air than the product made by method I.

III. Other preparative methods are the dry reduction of  $BeSO_4$  with agents containing no hydrogen, e.g., S vapor, CO,  $CaC_2$ , Zn or Al. The best yields are obtained with Al (97%, based on  $BeSO_4$ ). The reduction begins at 560°C.

Beryllium sulfide may also be obtained by reaction of  $CS_2$  with BeO at temperatures above 1200°C.

#### PROPERTIES:

Gray to white powder with a faint odor of  $H_2S$  when exposed to air. Crystallizes in structure type B3 (ZnS). d 2.36.

**REFERENCES:** 

- I. H. von Wartenberg, Z. anorg. allg. Chem. 252, 136 (1943).
- II. E. Tiede and F. Goldschmidt, Ber. dtsch. chem. Ges. <u>62</u>, 758 (1929).
- III. P. Silber, Ann. Chim. (12) 7, 182 (1952).

## Beryllium Selenide and Beryllium Telluride

BeSe, BeTe

BeSe

$$\begin{array}{rrrr} \mathbf{Be} &+& \mathbf{Se} &=& \mathbf{BeSe} \\ 9.0 && 79.0 && 88.0 \end{array}$$

Beryllium selenide is prepared from the elements in a  $H_2$  stream at 1100°C. Pure Se and pure pulverized Be are placed in a quartz reaction tube in separate boats made of  $Al_2O$  or BeO, (or at least of quartz). The hydrogen should pass first over the heated Se and then, when laden with its vapor, over the Be. A wash bottle filled with lead acetate is mounted at the exit end of the reaction tube to absorb the very toxic  $H_2$ Se present in the discharged gas. The Se is heated with a Bunsen burner; the uniform heat of an electric furnace is required for the Be. The BeSe so obtained often shows a tendency to crystallize in long needles.

ВеТе

Beryllium telluride can be prepared from the elements by the same method.

Be + Te = BeTe 9.0 127.6 136.6

PROPERTIES:

Gray powders, decomposing relatively rapidly in air. Crystallize in structure type B3 (ZnS type). d (BeSe) 4.32, d (BeTe) 5.09.

**REFERENCE:** 

Private communication from E. Tiede.

## **Beryllium Nitride**

Be<sub>3</sub>N<sub>2</sub>

 $3 \text{ Be} + 2 \text{ NH}_3 = \text{Be}_3 \text{N}_2 + 3 \text{H}_2$ 27.1 34.1 55.1 6.1

Instead of synthesizing the nitride from the elements, it may be obtained more easily and in higher yield by heating metallic Be in an  $NH_3$  stream.

The commercial metal is crushed in a steel mortar and screened through a 100-mesh (per inch) sieve; the powder is placed in a corundum boat, which is heated in a porcelain tube in a dry  $NH_3$  stream at 850°C for three hours. The reaction product is then pulverized in an agate mortar and heated in the  $NH_3$  stream at 1000°C; this procedure is repeated three times. The product so obtained usually contains only 94-95%  $Be_3N_2$ . A purer product might be obtained by the use of metal distilled in high vacuum and by careful manipulation.

PROPERTIES:

Gray-white powder; stable in air; decomposes rather slowly on contact with boiling acids. Crystallizes in structure type  $D5_3$  (C sesquioxide type).

**REFERENCES:** 

F. Fichter and E. Brunner, Z. anorg. allg. Chem. <u>93</u>, 86 (1915). J. Rieber, Thesis, Hannover, 1930.

M. von Stackelberg and R. Paulus, Z. phys. Chem. (B) 22, 305 (1933).

898

## **Beryllium Azide**

#### $Be(N_3)_2$

 $\frac{\text{Be}(\text{CH}_3)_2}{39.0} + \frac{2}{86.0} + \frac{2}{93.0} + \frac{2}{44.8} \frac{\text{CH}_4}{44.8 l}$ 

Pure dimethylberyllium is sublimed into a reaction vessel cooled with liquid  $N_2$ ; then an absolutely dry ether solution of excess  $HN_3$  is condensed on top of the dimethylberyllium layer. As the reaction mixture thaws, a vigorous reaction with evolution of methane starts even before all of the ether is melted (m.p.  $-116^{\circ}$ C), and Be( $N_3$ )<sub>2</sub> separates out as a white precipitate. The ether and excess  $HN_3$  are then distilled off under high vacuum.

PROPERTIES:

White, solid substance; explosive in the presence of a flame; insensitive to shock. Rapidly decomposed in moist air. Hydrolyzes in aqueous solution to such an extent that the substance may not be recovered undecomposed even when the solvent is very carefully distilled off in a high vacuum.

**REFERENCE:** 

E. Wiberg and H. Michaud, Z. Naturforsch. 9 b, 502 (1954).

## **Beryllium Carbides**

Be<sub>2</sub>C; BeC<sub>2</sub>

Be<sub>2</sub>C

I.

$$\begin{array}{rcrcrcc} 2 & Be & + & C & = & Be_2C \\ 18.0 & 12.0 & 30.1 \end{array}$$

A mixture of metallic Be (turnings or powder) and finely divided graphite or calcined acetylene black is reacted at  $1700^{\circ}$  for 20 to 30 minutes. The yield is 85%.

II.  $2 \operatorname{BeO} + 2 \operatorname{C} = \operatorname{Be}_2 \operatorname{C} + 2 \operatorname{CO}_{50.0 \quad 36.0 \quad 30.0 \quad 56.0}$ 

A blend of calcined BeO and half its weight of carbon made from sugar is mixed with 5% of starch and 15% of water, compressed into cylinders, dried, calcined, and finally heated in  $H_2$  at two

atmospheres gauge and  $1930^{\circ}$ C for 10 to 15 minutes; an 85-92% pure product is obtained as beautiful, brick-red crystals. The reaction does not start below 1700°C, but the carbide decomposes extensively above 2200°C. For this reason, the more convenient electric arc furnace process is not recommended. According to Messerknecht and Biltz, the simplest form of the arc process apparatus consists of a graphite crucible with a carbon electrode immersed in the reactant mixture (110v., 30 amp.).

Low concentration products may be enriched by treatment with hot dilute hydrochloric acid (however, a portion of the carbide is lost by decomposition). The excess C may thus be decanted and the BeO dissolved. In this manner, an enrichment to more than 95%Be<sub>2</sub>C is possible; the balance is BeO and free C.

PROPERTIES:

Yellow-red, finely crystalline powder, slowly decomposing in moist air. Crystallizes in structure type C1 (antifluorite type).

**REFERENCES:** 

- I. G. Oesterheld, Z. anorg. Chem. <u>97</u>, 1 (1916); M. W. Mallett, E. A. Durbin, M. C. Udy, D. A. Vaughan and E. A. Center, J. Electrochem. Soc. 101, 298 (1954).
- II. F. Fichter and E. Brunner, Z. anorg. Chem. 93, 91 (1915); J. Kielland and L. Tronstad, Kong. Norske Vidensk. Selsk. Forhandl. 8, 147 (1936); P. Lebeau, Comptes Rendus Hebd. Séances Acad. Sci. 121, 496 (1895); C. Messersknecht and W. Biltz, Z. anorg. allg. Chem. 148, 153 (1925); M. von Stakkelberg and F. Quatram, Z. phys. Chem. (B) 27, 50 (1934).

BeC<sub>2</sub>

According to Durand,  $BeC_2$  is formed by passing dry acetylene at 450°C over Be powder in a Pyrex tube. The product appears black due to free C formed via thermal decomposition of the acetylene.

PROPERTIES:

Generates acetylene with  $\mathrm{H_{2}O}$  and (moderately) with dilute HCl.

REFERENCE:

J. F. Durand, Bull. Soc. Chim. France [4] 35, 1141 (1924).

# Beryllium Acetate

## Be(CH<sub>3</sub>COO)<sub>2</sub>

 $\begin{array}{rl} \text{Be}_{4}\text{O}(\text{CH}_{3}\text{COO})_{6} &+ 2\ \text{CH}_{3}\text{COO} \cdot \text{Cl} &+ 2\ \text{CH}_{3}\text{COOH} \\ & \\ & 406.3 & 157.0 & 120.1 \\ & = 4\ \text{Be}(\text{CH}_{3}\text{COO})_{2} + 2\ \text{HCl} &+ (\text{CH}_{3}\text{CO})_{2}\text{O} \\ & \\ & 508.4 & 72.9 & 102.1 \end{array}$ 

Basic beryllium acetate (4 g.) is dissolved in 50 ml. of boiling glacial acetic acid and refluxed with 4-5 g. of acetyl chloride for a short time. The precipitate of  $Be(CH_3COO)_2$  forming after a few minutes is filtered off, washed with glacial acetic acid and with cold chloroform, and dried in a vacuum desiccator. The yield is 90-94%.

**PROPERTIES:** 

Stable for several weeks in a closed vessel at room temperature. Slowly splits off acetic anhydride (rapidly when heated) and is converted to basic beryllium acetate, which sublimes. Undergoes partial decomposition to acetic anhydride and beryllium oxide when heated rapidly.

Barely attacked by cold water, is hydrated at elevated temperature. Insoluble in all solvents for basic beryllium acetate.

**REFERENCE:** 

J. Besson and H. D. Hardt, Z. anorg. allg. Chem. 277, 188 (1954).

## **Basic Beryllium Acetate**

#### Be<sub>4</sub>O(CH<sub>3</sub>COO)<sub>6</sub>

I. Basic beryllium carbonate (40 g.) is mixed with 80 ml. of glacial acetic acid and stirred, with heating, until  $CO_2$  evolution ceases. The end of the reaction is also recognized by the start of precipitation of white, semitranslucent crystals; otherwise, there is an amorphous, white residue. The solution is cooled to room temperature and the crystallized basic acetate is filtered off and dried in air.

The crude product is treated with 60-80 ml. of chloroform, and any insoluble residue is removed by filtration. The basic salt, which crystallizes in colorless octahedra, is filtered off and freed of residual chloroform in a vacuum desiccator. The product melts at  $284^{\circ}$ C and sublimes without a residue at reduced pressure. The yield is 28 g. II. A method described by Hardt avoids contamination of the product by ammonium salts and solvent occlusions and makes purification by sublimation unnecessary.

Beryllium hydroxide or basic beryllium carbonate is stirred with glacial acetic acid to a paste and evaporated to dryness in a laboratory oven at 120 to  $130^{\circ}$ C. The crude product thus formed is extracted with glacial acetic acid in a Soxhlet apparatus (using a glass wool filter); after cooling, it crystallizes from the extract in well-formed octahedra, which are filtered off and dried at  $130^{\circ}$ C.

This is followed by a second Soxhlet extraction with  $CCl_4$ , since the ammonium salts from the starting material still remain undissolved. The product is analytically pure after drying at  $100^{\circ}C$ .

PROPERTIES:

At room temperature the molecular lattice is of the space group  $T_n^4(\alpha)$ . Transition to other modifications  $(\gamma, \beta)$  occurs at 150 to 155°C. Very soluble in chloroform; readily soluble in boiling benzene, toluene, xylene, Tetralin and glacial acetic acid; less soluble in CCl<sub>4</sub>, acetic anhydride and acetyl chloride; sparingly soluble (0.3%) in diethyl ether.

In anhydrous boiling methanol following initial dissolution, splits off acetic anhydride with formation of highly aggregated basic acetates.

Quite stable in cold water, but is rapidly hydrolyzed in hot water.

Begins to sublime at about  $200^{\circ}$ C, softens at about  $280^{\circ}$ C to a nematic or smectic (liquid crystal) state and melts at 183 to  $184^{\circ}$ C with sublimation. May be distilled at 1-2 atmospheres gauge. Pure basic beryllium acetate leaves a residue of 0.3-0.5% BeO after sublimation.

For the preparation of basic beryllium formate and basic beryllium propionate, see the literature cited under II; also H. Hendus and H. D. Hardt, Z. anorg. allg. Chem. 277, 127 (1954).

REFERENCES:

- F. Haber and G. van Oordt, Z. anorg. allg. Chem. 40, 465 (1904).
- A. Stock, P. Praetorius and O. Priess, Ber. dtsch. chem. Ges. <u>58</u>, 1571 (1925). Übungsbeispiele aus der anorg. Experimentalchemie[Practical Experiments in Inorganic Chemistry], Leipzig 1920, p. 212.
  - I. T. Moeller, A.J. Cohen and E. Marvell in: L.F. Audrieth, Inorganic Syntheses, Vol. III, New York-Toronto-London 1950, p. 9. p. 9.
- II. J. Besson and H. D. Hardt, Z. anorg. allg. Chem. 277, 188 (1954); H. Hendus and H. D. Hardt, Z. anorg. allg. Chem. 286, 265 (1956); H. D. Hardt, Z. anorg. allg. Chem. 286, 254 (1956).

## Magnesium

#### Mg

Generally, very pure magnesium is prepared by refining commercial magnesium via distillation or, still better, by sublimation in high vacuum.

Magnesium is usually made commercially by electrolysis of molten, dehydrated carnallite (MgCl<sub>2</sub>  $\cdot$  KCl), with an Acheson graphite anode and an iron cathode.

The crude metal obtained by electrolysis contains up to 2.7% Cl, besides other impurities (Fe, Al, Si, N). Purification is effected by melting with fluxing agents or, better, by repeated filtration. This may be done simply by pushing a sheet iron screen downward through the melt. This reduces the chlorine content considerably (final content: 0.003% Cl).

Another method of purification consists in refining with Zr. Addition of 2-4% of  $ZrCl_4$  to Mg melts results in precipitation of Fe, Al, Si and Mn but not of Cu. The Zr can be removed by subsequent passage of  $H_2$ , still retaining the degree of purity previously achieved. Magnesium thus prepared is highly resistant to corrosion.

Pure magnesium (99.9%), absolutely free of halogen, is obtained by the Radentheiner process.

## I. SUBLIMATION OF THE CRUDE METAL

In the laboratory, the following simplified method may be used: the sublimation vessel shown in Fig. 255 (see also the apparatus described for the distillation of Ca) consists of an iron tube (60-mm. diameter and 500 mm. long), which is closed at one end and has a threaded vacuum-tight lid on the other. The lid is fitted with a nipple for connection to a high-vacuum system. The lower end of this retort contains an iron tube f with the crude magnesium; the open end of the tube is preferably closed by one or two finemesh wire screens which are clamped to the tube. The conical iron tube e is inserted into the colder zone of the sublimation vessel. The tube consists of two halves held together by rings (see Fig. 255). Before the preparation, the interior walls of the tube are dusted with MgO in order to prevent sticking of the condensate.

The sublimation temperature is stated in the literature to be about  $600^{\circ}$ C at a vacuum of  $10^{-3}$  mm. If the condensing surface is at 400 to 500°C, the crystals will be dendritic; larger quantities agglomerate to massive blocks. At higher temperatures ( $600^{\circ}$ C), large single crystals are formed.

The metal obtained after a single sublimation contains Fe, Al, Si and Cl in amounts less than 0.001%. After repeated sublimation the impurities cannot be identified either chemically or spectroscopically. The sublimed metal can be remelted in an argon atmosphere at 300-400 mm.



Fig. 255. Purification of magnesium by sublimation.

II. When larger quantities (about 1 kg.) must be sublimed in a single operation, the metal is condensed in an apparatus of larger capacity, using a water-cooled cylinder.

PROPERTIES:

Atomic weight 24.32. M.p. 650°C, b.p. 1102°C; d 1.737. Crystallizes in structure type A3.

Cold water reacts only slowly with magnesium; the reaction is much more rapid in boiling water. The metal dissolves violently in acids.

REFERENCES:

Gmelin, Handb. d. anorgan. Chemie, 8th ed., Vol. 27B (Magnesium), p. 121.

- H. Funk, Die Darstellung der Metalle im Laboratorium [Preparation of Metals in the Laboratory] Stuttgart 1938, p. 26.
- K. E. Mann, Z. Metallkunde 44, 264 (1953).
- I. W. Kaufmann and P. Siedler, Z. Elektrochem. 37, 492 (1931).
- II. I. Hérenguel and G. Chaudron, Comptes Rendus Hebd. Séances Acad. Sci. <u>193</u>, 771 (1931); <u>195</u>, 1272 (1932); see also G. Chaudron in A. E. van Arkel, Reine Metalle [Pure Metals], Berlin 1939, p. 111.

## Magnesium Hydride MgH<sub>2</sub>

I.

 $Mg + H_2 = MgH_2$ 24.3 22.4 *l* 26.3

Magnesium and hydrogen react at  $570^{\circ}C$  and 200 atmospheres in the presence of MgI<sub>2</sub> as a hydrogen transfer agent to form MgH<sub>2</sub>. The yield is 60%.

II. Magnesium hydride is formed when magnesium dialkyls (diethyl, dibutyl, diphenyl) or the corresponding Grignard compounds are heated to 175-200°C under high vacuum for several hours.

PROPERTIES:

White, water-sensitive solid; not spontaneously combustible. Insoluble in ether; nonvolatile; stable to  $280^{\circ}$ C under vacuum.

**REFERENCES:** 

- I. E. Wiberg, H. Goeltzer and R. Bauer, Z. Naturforsch. <u>6</u> b, 394 (1951); see also F. Ellinger et al. J. Amer. Chem. Soc. <u>77</u>, 2647 (1955).
- II. E. Wiberg and R. Bauer, Z. Naturforsch. <u>5</u> b, 396 (1950); E. Wiberg and R. Bauer, Chem. Ber. 85, 593 (1952).

## **Magnesium Chloride**

MgCl<sub>2</sub>

#### ANHYDROUS MgCl<sub>2</sub>

In the method developed by Richards,  $MgCl_2$  is prepared from a mixture of  $NH_4Cl + MgCl \cdot 6H_2O$  or from the double salt  $NH_4Cl \cdot MgCl_2 \cdot 6H_2O$  by dehydration in a HCl stream. The reaction of MgO with a gas mixture of CO and  $Cl_2$  is much better suited for the preparation of larger quantities (more than 100 g.).

I.

 $\begin{array}{rl} NH_4Cl \ + \ MgCl_2 \cdot 6 \ H_2O \ \ or \\ 53.5 & 203.3 \\ NH_4Cl \ \cdot \ MgCl_2 \cdot 6 \ H_2O \ = \ MgCl_2 \ + \ NH_4Cl \ + \ 6 \ H_2O \\ 256.8 & 95.2 & 53.5 & 108.1 \end{array}$ 

Equimolar quantities of  $MgCl_2 \cdot 6H_2O$  and  $NH_4Cl$  are crystallized from aqueous solution somewhat above 50°C. After separating the mother liquor at this temperature the salt is recrystallized once, slightly predried and filled while still hot into a quartz boat, which is then inserted into a quartz tube. Dehydration in a dry HCl stream proceeds first at 100°C for three hours (avoid melting of the hydrated crystals), then at 250°C for one hour, and finally at 400°C for one hour. Finally, the product is quickly melted and freed of HCl while cooling in a stream of CO<sub>2</sub>. The salt thus prepared contains about 0.1% MgO and 0.05% SiO<sub>2</sub>; it may be stored over  $P_2O_5$ .

If high purity is not required, it will suffice to start with a mixture of  $NH_4Cl$  and  $MgCl_2 \cdot 6H_2O$ .

The consumption of HCl may be reduced substantially through dehydration of the hexahydrate or of ammonium carnallite by heating at 200°C under vacuum or by storage in a desiccator over  $P_2O_5$ . The drying with HCl may then be started directly at 200°C.

Even though HCl begins to split off at  $106^{\circ}$ C when the hexahydrate is heated in air, it is possible to dry the material at  $150^{\circ}$ C without formation of undesirable amounts of basic salt, provided the evaporation is rapid. It is advisable to place the hexahydrate in a large porcelain dish covered with a wide funnel, which serves as a partial condenser of the hydrochloric acid vapors formed. Thus, the formation of the basic salt is reduced to a minimum. The dihydrate begins to crystallize at  $180^{\circ}$ C. At this point the melt is poured onto an aluminum sheet and crushed while still warm. This product still contains 2.3-2.8 moles of water per mole of MgCl<sub>2</sub>.

According to Treadwell, the HCl may be recycled after drying with concentrated  $H_2 SO_4$ , but this requires elaborate equipment; besides, traces of chlorosulfonic acid formed react with MgCl<sub>2</sub> to yield MgSO<sub>4</sub>.

п.

Magnesium oxide can be chlorinated quantitatively only in the presence of a reducing agent (CO). The reaction depends critically on the thermal pretreatment of the MgO; magnesium carbonate calcined at  $800^{\circ}$ C is still very reactive. Commercial CO in cylinders is not suitable, since the H<sub>2</sub>it contains forms water, which damages the apparatus; for this application CO is prepared in a small generator.

a. Carbon monoxide generator. The reactor is a vertically mounted quartz tube with an I.D. of 20 mm. and length of 600 mm., which is maintained at about  $950^{\circ}$ C. A somewhat narrower, sealed-off quartz tube is pushed into the lower hot zone; this tube serves as a retaining grating. The reactor is charged with dry activated carbon of 5-mm. particle size and may be refilled through a side arm sealed on at the upper end. To start the reaction,  $CO_2$  from a cylinder is added from the top. A small CO flame is allowed to burn at a capillary branching off from the outlet tube to provide a simple test for the CO content of the generated gas. The outlet gas passes through a pressure relief valve and a bubble counter and is then mixed with  $Cl_2$ , which is metered by the same method.

furnace. b. Magnesium oxide The reaction vessel is a vertically mounted, 700-mm.-long quartz tube with an I. D. of 35 mm. In the lower portion there is a 4-mm.-diameter, funnelshaped, centered drip nozzle filled with a layer of quartz fragments to support a charge of 500 g. of MgO. The gas flow rate is 400 ml./ min. (200 ml.  $Cl_2$  + 200 ml. CO), and the reaction temperature is 750°C. The exit gas has a 1% excess of CO and this ensures that the Cl<sub>2</sub> content is kept below  $3 \cdot 10^{-5}$ %. The gases flow upward, countercurrent to the salt, which drips down. The salt drops emerging from the nozzle should fall free into the receiver which is mounted underneath by means of a ground glass joint. The pure white, brittle rods and grains of MgCl<sub>2</sub> may be easily removed later. However, the entire oxide charge must be wetted with molten chloride before the first drops appear. This occurs only 2-3 hours after the start of the chlorination.

The exit gases from the top of the reactor pass a small dust collector, which retains, aside from the dust, the impurities of the oxide. The gases are then vented. Through a capillary branching off the gas discharge line, a small gas stream may be diverted to a Bunsen flame containing a glowing copper wire. In this way one may test the gas composition. The desired excess of CO can be recognized by a faintly blue cast; traces of Cl give a green copper flame (Beilstein test).

#### PROPERTIES:

Leafy crystalline mass. Melts at  $712^{\circ}$ C to a water-clear, mobile liquid. May be distilled at bright red heat in a H<sub>2</sub> stream. d 2.41. More hygroscopic than the chlorides of the higher alkaline earth metals. Liberates Cl<sub>2</sub> when heated to  $300^{\circ}$ C forming oxides and oxychlorides.

**REFERENCE:** 

I. T. W. Richards and H. G. Parker, Z. anorg. Chem. <u>13</u>, 81 (1897); W. Biltz and G. F. Hüttig, Z. anorg. allg. Chem. <u>119</u>, 116 (1921); W. Biltz and W. Klemm, Z. phys. Chem. <u>110</u>, 331 (1924); W. D. Treadwell and T. Zürrer, Helv. Chim. Acta <u>15</u>, 1271 (1932); K. K. Kelley and G. E. Moore, J. Amer. Chem.

Soc. <u>65</u>, 1264 (1943); J. S. Peake and W. L. Fielder, Proc. Indian Acad. Sci. 63, 113 (1953).

II. W. D. Treadwell, A. Cohen and T. Zürrer, Helv. Chim. Acta 22, 449 (1939).

## MgCl<sub>2</sub> · 6 H<sub>2</sub>O

Very pure material may be prepared from 500 g. of commercial  $MgCl_2 \cdot 6 H_2O$ , which is dissolved in water. The solution is saturated with  $H_2S$ , some ammonia is added, and the solution is kept warm for several days. The supernatant solution is decanted and Ca is precipitated from it by the addition of a small amount of very pure ammonium oxalate. After settling, the decanting is repeated. The filtrate is tested for completion of the reaction by repeated addition of oxalate, followed by long settling.

The clear solution is evaporated and the salt is calcined. The mixture of MgO and oxychloride thus obtained is washed on a filter for 60 hours with distilled water until the wash water no longer contains any Na and K. Nevertheless, the Mg salt redissolved in pure distilled hydrochloric acid may not be free of these metals; the procedure described above must then be repeated until no further traces of Na and K can be detected.

On concentration, the MgCl<sub>2</sub> ° 6  $H_2O$  precipitates from the aqueous solution between -3 and +116°C.

Quartz should be used if no platinum equipment is available. It is advisable to use electric heating to avoid contamination by the gases of the flame (sulfur).

## PROPERTIES:

The hexahydrate forms bitter, deliquescent, monoclinic crystals. d 1.56.

## AMMONIUM CARNALLITE

The ammonium chloride required for the preparation of the double salt is treated with nitric acid to break down any amines and is then dried and sublimed several times. It is then recrystallized 5 to 6 times, and finally repeatedly sublimed in a stream of pure air (which is washed with concentrated  $K_2CO_3$  and concentrated  $H_2SO_4$ ).

The pure double salt is crystallized from a mixture of one mole of  $MgCl_2 \cdot 6 H_2O$  in 30 ml. of water and one mole of  $NH_4Cl$  in 70 ml. of water above 50°C (preferably after boiling for a few minutes).

REFERENCES:

T. W. Richards and H. G. Parker, Z. anorg. Chem. 13, 81 (1897).

E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 89. For the preparation of the tetra-, di- and monohydrates of MgCl<sub>2</sub> see: C. H. Shomate and E. H. Huffman, J. Amer. Chem. Soc. 65, 1625 (1943); see also R. Manocha and G. Sternheim, J. Sci. Ind. Res. 15 B, 375 (1956).

#### **Magnesium Bromide**

#### MgBr<sub>2</sub>

Magnesium bromide can be obtained by dehydration of the hexahydrate in a stream of HBr (I) or via the reaction of  $Br_2$  with Mg in anhydrous diethyl ether (II).

I.  $\begin{array}{ccc} MgCl_2 \cdot 6 \ H_2O \rightarrow MgBr_2 \cdot 6 \ H_2O \rightarrow MgBr_2 \\ 203.3 \ & 292.2 \ & 184.1 \end{array}$ 

The hydroxide is precipitated with ammonia from  $MgCl_2$  solution. It is then washed until the dissolved sample is halogen-free, suspended in water, and dissolved by the introduction of HBr gas. The salt obtained by concentrating the solution is recrystallized and dehydrated in a stream of HBr. Traces of HBr are removed by heating in dry, oxygen-free N<sub>2</sub>.

If higher purity material is desired, quartz equipment must be used. Ground glass joint apparatus prevents contact of the salt with the atmosphere. The procedure follows the Baxter method for the preparation of alkaline earth halides.

П.	Mg	$+ Br_2 =$	MgBr
	24.3	159.8	184,1

Pure  $Br_2$  is evaporated at 50 to  $55^{\circ}C$  in a wash bottle and carried by a stream of dry N<sub>2</sub> (15 ml./minute) to the bottom of a round, 500-ml. flask. The flask contains 150 ml. of freshly distilled, anhydrous ether and 10 g. of clean Mg turnings. It is equipped with a high-speed stirrer and a CaCl<sub>2</sub> tube on the gas outlet tube (to exclude atmospheric moisture). Externally the flask is cooled to below room temperature to remove the considerable heat of reaction.

The reaction is stopped when 15 ml. of  $Br_2$  has evaporated. The liquid is decanted into a dry flask and crystallization of the MgBr<sub>2</sub> trietherate is initiated by cooling below 0°C. The mother liquor is discarded. The crystals are immediately treated with anhydrous benzene and allowed to stand at room temperature. The ether-benzene mixture dissolves most of the impurities, while MgBr<sub>2</sub> is only slightly soluble in it. The suspension is then again cooled to 0°C; the crystals are rapidly filtered off and washed with cold (0°C) benzene. The ether is removed by a water jet aspirator and the etherate is decomposed by raising the temperature to  $150-175^{\circ}$ C over a period of a few hours. Oil-pump vacuum is applied for one hour to remove residual ether. The yield of MgBr<sub>2</sub> is 60-70%. The salt is 99.3-99.6% pure; without the benzene purification step, the purity is only 90-95%.

PROPERTIES:

Pure white salt. M.p. 711°C; d 3.72. Crystallizes in structure type C6. Very hygroscopic. The melt is decomposed by atmospheric oxygen and turns yellow, forming MgO and Br<sub>2</sub>.

PROPERTIES:

- I. M. Linhard and M. Stephan, Z. phys. Chem. (A) <u>167</u>, 88 (1933);
   G. F. Baxter and F. N. Brink, J. Amer. Chem. Soc. <u>30</u>, 47 (1908).
- H. H. Rowley, J. Amer. Chem. Soc. 72, 3305 (1950); B. Menschutkin, Z. anorg. Chem. 49, 40 (1906); W. Biltz and G. F. Huttig, Z. anorg. allg. Chem. 119, 115 (1921).

#### **Magnesium** Iodide

#### Mgl<sub>2</sub>

Magnesium iodide can be prepared either by dehydration of its hydrate in an HI stream (I), from the elements (II), or via the reaction of  $I_2$  and Mg in anhydrous ether (III).

I. 
$$\begin{array}{ccc} MgI_2 \cdot 8 H_2O & \text{or} & MgI_2 \cdot 6 H_2O \rightarrow MgI_2 \\ 422.3 & 386.3 & 278.2 \end{array}$$

The method is analogous to method (I) for  $MgBr_2$ .

$$\begin{array}{cccc} \Pi_{\bullet} & & Mg + I_2 = MgI_2 \\ & & 24.3 & 253.9 & 278.2 \end{array}$$

Clean Mg turnings (0.5 g.) are heated to  $600^{\circ}$ C in a porcelain boat located in the center of a Vycor tube. The tube is evacuated and 5-6 g. of I<sub>2</sub> is placed at one end. The I<sub>2</sub> is sublimed repeatedly from one end of the tube to the other, thus passing over the Mg. Finally, the center only of the tube is heated. The I<sub>2</sub> then collects at the cold ends of the tube, while some Mg remains in the boat. The pure, white MgI<sub>2</sub> platelets deposit on either side of the boat. The tube must be cut open to remove the sublimate. III. Anhydrous  $MgI_2$  can also be obtained by decomposing the etherate under high vacuum at 230°C; the etherate may be prepared in the same way as described for  $MgBr_2$  (Method II). The method does not seem to offer any special advantage.

PROPERTIES:

Hexagonal platelets. M.p. in hydrogen  $650^{\circ}$ C; d 4.43. Crystallizes in structure type C6 (layer lattice). Extremely hygroscopic. At room temperature, crystals of the octahydrate precipitate from aqueous solution; the hexahydrate is stable above  $34^{\circ}$ C.

**REFERENCES:** 

- I. M. Linhard and M. Stephan, Z. physik. Chem. (A) <u>167</u>, 88 (1933).
- II, III. W. Blitz and G. F. Hüttig, Z. anorg. allg. Chem. <u>119</u>, 115 (1921); W. Klemm, K. Beyersdorfer and J. Oryschkewitsch, Z. anorg. Chem. 256, 25 (1948).

## **Magnesium Oxide**

#### MgO

Magnesium oxide is generally obtained by calcining easily decomposed magnesium compounds such as the hydroxide, the nitrate, the oxalate and other salts of organic acids. However, the principal raw material is the carbonate (the sulfate may also be converted to the oxide at high temperatures).

Basic magnesium carbonate is solidly packed into the lower half of a tall porcelain crucible. An electric crucible furnace is slowly heated. All the  $CO_2$  is removed within about one hour at  $600^{\circ}C$ . The still hot crucible is rapidly cooled over  $P_2O_5$  in a desiccator. The MgO is obtained as a loose, white mass which is readily soluble in dilute hydrochloric acid.

PROPERTIES:

Formula weight 40.32. M.p.  $2642^{\circ}$ C, b.p.  $2800^{\circ}$ C. Crystallizes in structure type B1; d 3.58. The physical properties and the chemical reactivity depend to a large extent on the method of preparation (starting material, calcining temperature, time).

REFERENCES:

Private Communication from E. Tiede.

Gmelin, Handb. der anorg. Chemie, 8th ed., Vol. 27 B (Magnesium) p. 12.
# Magnesium Hydroxide Mg(OH)<sub>2</sub>

Magnesium hydroxide may be precipitated with alkali from solutions of magnesium salts. It may also be obtained by hydration of magnesium oxide and by the reaction of water with magnesium amalgams.

I. MICROCRYSTALLINE Mg(OH)2:

 $\begin{array}{rl} MgCl_{2} \cdot 6 \ H_{2}O \ + \ 2 \ NH_{4}OH \ = \ Mg(OH)_{2} \ + \ 2 \ NH_{4}Cl \ + \ 6 \ H_{2}O \\ 203.3 \ & 70.1 \ & 58.3 \ & 107.0 \ & 108.1 \end{array}$ 

A solution of  $MgCl_2 \cdot 6 H_2O$ , almost saturated at room temperature, is treated with a large excess of concentrated ammonium hydroxide at 70°C, while high-speed agitation is maintained. Since  $Mg(OH)_2$  is very sensitive to  $CO_2$ , the distilled water employed must be thoroughly boiled before use. The ammonia used for the precipitation is prepared by passing  $NH_3$  gas, from which  $CO_2$  has been carefully removed with solid KOH, into  $CO_2$ -free water. Because of the sensitivity of  $Mg(OH)_2$  to silicic acid, only Pyrex glass vessels should be used.

After the precipitation, the material should remain submerged under the mother liquor for two days. During this time it is reheated to  $70^{\circ}$ C several times. The precipitate is purified first by washing with CO<sub>2</sub>-free water, then by centrifuging in paraffincoated nickel tubes. Precipitation and washing are carried out in the apparatus shown in Fig. 256, which is self-explanatory. The substance is dried under vacuum, first over solid KOH and then over P<sub>2</sub>O<sub>5</sub>.



Fig. 256. Preparation of magnesium hydroxide.

The impurities in the material thus prepared are approximately: < 0.1% Cl, 0.27% CO<sub>2</sub>, 0.01% SiO<sub>2</sub>. Loss on calcining (1000°) 31.36% (theoretical: 30.88%).

## II. MACROCRYSTALLINE Mg(OH)2:

 $\begin{array}{rl} MgCl_2 \cdot 6 \ H_2O \ + \ 2 \ KOH \ = \ Mg(OH)_2 \ + \ 2 \ KCl \ + \ 6 \ H_2O \\ 203.3 & 112.2 & 58.3 & 149.1 & 106.1 \end{array}$ 

A wide-neck Pt vessel is charged with 1375 g. of KOH, 48.5 g. of MgCl<sub>2</sub>.6 H<sub>2</sub>O and 243 ml. of water. This mixture is heated in an electric furnace to  $210^{\circ}$ C and held at this temperature until the melt becomes clear (about 30 minutes). It is then allowed to cool to room temperature over a period of 18 hours. The solidified melt is dissolved in water and the solution filtered through a glass frit. The crystals remaining on the filter are washed several times with distilled water and dried at  $100^{\circ}$ C for two hours. The well-shaped crystals have a diameter of about 0.2 mm; they are free of CO<sub>2</sub> and contain about 0.15% K.

#### PROPERTIES:

d 2.4. Crystallizes in structure type C6. Readily soluble in acids. Thermal degradation to MgO begins at temperatures above  $200^{\circ}$ C under vacuum.

#### **REFERENCES:**

- I. R. Fricke, R. Schnabel and K. Beck, Z. Elektrochem. <u>42</u>, 881 (1936); for other preparative methods, see there, as well as R. Fricke et al., Z. Elektrochem. <u>41</u>, 174 (1935); Z. anorg. allg. Chem. 166, 255 (1927).
- II. A. deSchulten, Comptes Rendus, Hebd. Séances Acad. Sci. 101, 72 (1885); W. F. Giauque and R. C. Archibald, J. Amer. Chem. Soc. 59, 561 (1937).

### Magnesium Sulfide

### MgS

The synthesis of MgS from its elements is impractical since the reaction can be very violent. Heating of the metal in a stream of  $H_2S$  is more adaptable to the laboratory. Very pure MgS is formed in the reaction of  $CS_2$  with MgSO<sub>4</sub> at 900°C. If one prefers not to use an oxygen-containing compound as a starting material, ammonium magnesium chloride may be reacted in a stream of  $H_2S$ .

I. 
$$Mg + H_2S = MgS + H_2$$
  
24.3 34.1 56.4 2.0

A stream of dry  $H_2S$  is passed over Mg turnings placed in a graphite boat or, better, in a boat made of sintered magnesia ( $H_2S$  flow rate 8 ml./min.). The reaction starts at 580°C, and once started, may be continued at lower temperature and higher flow rate (15 ml./min.). Unreacted Mg is distilled off by heating to 800°C under high vacuum. The pure white product contains 99.5% MgS; it reacts vigorously with water at room temperature.

II.  $3 MgSO_4 + 4 CS_2 = 3 MgS + 4 COS + 4 SO_2$ 361.1 304.6 169.2 240.3 256.2

The Von Wartenberg modification of the CS<sub>2</sub> process of Tiede and Richter proceeds as follows: about 20 g. of finely pulverized MgSO<sub>4</sub> (evaporated with a small excess of  $H_2SO_4$ ) is placed in a quartz tube just before all of the  $H_2SO_4$  has been removed. The sulfate is heated in pure N<sub>2</sub> for half an hour at 700°C and then in N<sub>2</sub> saturated with CS<sub>2</sub> vapor for ten hours at 750°C. At this point, the iodine solution test should show no further SO<sub>2</sub> in the exit gas. The reaction temperature cannot be raised since elemental C begins to separate from the gas at 800°C. The very pure, loose product thus obtained is completely white.

III.  $NH_4Cl \cdot MgCl_2 \cdot 6H_2O + H_2S = MgS + 2HCl + NH_4Cl + 6H_2O$ 256.8 34.1 56.4 72.9 53.5 108.1

In the Banks modification of the Sarge method, the reaction temperature must be held below the melting point of  $MgCl_2$  for about one hour while  $H_2S$  is passed over ammonium carnallite. It is then gradually (90 minutes) raised to  $1000^{\circ}C$ . The product is a white to cream-colored powder, which is very pure except for traces of chloride.

PROPERTIES:

White powder. M.p.  $> 2000^{\circ}$ C; d 2.86. Crystallizes in structure type B1. Crystalline MgS is only slowly attacked by water; it is completely soluble in dilute HCl.

REFERENCE:

- I. K. Nielsen, Ann. Chim. [12] 2, 354 (1947).
- II. E. Tiede and F. Richter, Ber. dtsch. chem. Ges. 55, 69 (1922);
   H. von Wartenberg, Z. anorg. allg. Chem. 252, 136 (1934).
- III. T.W.Sarge, US Pat. 2358661, Sept. 19, 1944; E. Banks, V.J.Russo and R. Ward, J. Amer. Chem. Soc. 72, 3173 (1950).

914

# **Magnesium Selenide**

## MgSe

The preparation of MgSe is analogous to that of BeSe:

Mg + Se = MgSe24.3 79.0 103.3

The optimal reaction temperature is  $750^{\circ}$ C (see also MgTe below).

PROPERTIES:

Slightly gray powder; decomposes very rapidly in air. Crystallizes in structure type B1. d 4.21.

REFERENCES:

Private communication from E. Tiede.

## **Magnesium Telluride**

#### MgTe

The direct preparation from the elements proceeds very violently at elevated temperature; it can be controlled by reacting only small quantities at a time.

Commercial Mg of at least 99.8% purity is used; the tellurium must be purified in most cases because of its selenium content. Twice recrystallized basic tellurium nitrate is freed of nitric acid by boiling in concentrated  $H_2SO_4$ , and then HCl gas is passed through the boiling solution for several hours. Finally, the solution is diluted and the Te is precipitated with hydrazine. The Te, still containing some oxide, is distilled under vacuum.

The apparatus for the synthesis of MgTe consists of a tube with male ground joints at both ends. The tube is fitted with a side arm into which a small funnel can be inserted; the funnel may be closed by means of a glass rod. A small amount of a fine mixture of Mg powder and Te (atomic ratio 1:1.1 to 1:1.2) is added to the funnel and, by lifting the glass rod, dropped into a boat made of corundum or preferably, of carbon (MgO may also be suitable). A hydrogen flow is maintained over the boat during the filling. The boat is then pushed sideways and the contents made to react by fanning with a small flame. This operation is repeated until a sufficient amount has accumulated. A small electric oven is then placed over the

reactor tube, one end and the side arm are closed with ground-glass caps, and the other end is connected to a high-vacuum system. The substance is then heated under high vacuum at 600 to  $700^{\circ}$ C for a long time in order to distill off excess tellurium. After the completion of this heat treatment, the cooled product is bottled under highly purified nitrogen.

PROPERTIES:

Pure white powder. Decomposes in damp air, forming  $H_2$ Te, which then oxidizes to Te and appears on the surface of the telluride as a black deposit. d 3.85. Crystallizes in structure type B4 (wurtzite type).

**REFERENCE:** 

W. Klemm and K. Wahl, Z. anorg. allg. Chem. 266, 289 (1951).

# Magnesium Nitride

# $Mg_3N_2$

Finely divided Mg reacts with nitrogen at elevated temperatures. The nitrogen must be very pure in order to obtain oxide-free nitride. For this reason, using a dry  $NH_3$  stream, instead of the  $N_2$ , is advisable.

Magnesium filings are placed in a boat made of porcelain or, preferably, of sintered magnesia and inserted into a porcelain tube. This tube is connected by means of a tee with a source of  $N_2$  and an apparatus for generating dry  $NH_3$ . The other end of the porcelain tube is connected to a U tube filled with equal volumes of CaO and KOH pellets. The exit gases are passed through an absorption unit consisting of two Erlenmeyer flasks filled with dilute  $H_2SO_4$ . The inlet tube of the first flask does not dip into the liquid.

After the air is completely displaced from the apparatus by the  $NH_3$  (air bubbles cease to emerge from the second Erlenmeyer flask), the Mg is heated at 800 to  $850^{\circ}C$  for four hours. The onset of nitride formation is recognized by incandescence of the Mg and the evolution of  $H_2$ . A high  $NH_3$  flow must be maintained at the peak of the reaction to avoid sucking the absorption fluid into the reactor. Since the finished material always contains adsorbed  $NH_3$ , heating in a  $N_2$  stream should be continued at the same temperature for 90 minutes. Because of the high moisture sensitivity of the material, bottling must be carried out with the usual precautions.

PROPERTIES:

Loose powder, green-yellow to yellow-orange; d 271. Crystallizes in structure type  $D5_3$  (carbon sesquioxide type). Very sensitive to moisture; decomposes rapidly in air to  $Mg(OH)_2$  and  $NH_3$ .

REFERENCES:

J. Rieber, Thesis, Hannover 1930.

M. von Stackelberg and R. Paulus, Z. phys. Chem. (B) 22, 305 (1933); see also H. Grubitsch, Anorganisch-präparative Chemie [Inorganic Preparative Chemistry], Vienna, 1950, p. 306.

## **Magnesium Azide**

 $Mg(N_3)_2$ 

The preparation is analogous to that described for beryllium azide (p. 899), using diethylmagnesium in ether-dioxane. The reaction starts on thawing below  $0^{\circ}$ C.

PROPERTIES:

White substance, sensitive to moisture, insoluble in ether and in tetrahydrofuran. Slightly explosive on contact with a flame. Only the basic azide is recovered from aqueous solutions when the water is distilled off under high vacuum.

**REFERENCE:** 

E. Wiberg and H. Michaud, Z. Naturforsch. 9 b, 501 (1954).

# Magnesium Phosphide and Magnesium Arsenide Mg<sub>3</sub>P<sub>2</sub>, Mg<sub>3</sub>As<sub>2</sub>

3 Mg +	⊢2 <b>P</b> =	$= Mg_{3}P_{2}$	$3 \mathrm{Mg}$ +	2 As	$= Mg_{3}As_{2}$
73,0	61,9	134,9	73,0	149,8	222,8

I. In the method of Zintl and Husemann, a  $H_2$  stream laden with vapors of P or As is passed over heated pulverized magnesium.

The substances may be prepared in the apparatus shown in Fig. 257 without coming into contact with air. Here  $S_1$  and  $S_2$  are two Vycor boats;  $S_1$  is filled with 5 g. of purified red P (or 12 g. of sublimed As);  $S_2$  is filled with 4 g. of Mg powder, prepared from pure metal with a milling machine in the absence of air. After thorough evacuation, pure  $H_2$  is introduced at H and escapes at A. Boats  $S_1$  and  $S_2$  are separately heated with two electric furnaces  $O_1$  and  $\mathcal{O}_{\mathbf{2}}$ . The initial temperature is 600°C in both furnaces. At the end of the reaction, the product in  $S_{\mathbf{2}}$  is kept at about 700°C for a short time in order to remove the excess of P (or As). The Vycor cylinder Z, which is sealed at one end, prevents the back-diffusion of the P (or As) vapor.

This method yields the compounds in finely crystalline form; the Mg<sub>3</sub> P<sub>2</sub> is bright yellow, the Mg<sub>3</sub> As<sub>2</sub>, intensely brown-red. The cylinder Z with the two boats is pushed toward B (against the H<sub>2</sub> flow) by means of a glass rod introduced at A. Boat  $S_2$  is then pulled to the left to C and overturned at that point so that the material accumulates at  $D_{\bullet}$ . Outlet A is then closed off and the compound is pulverized at D by means of a glass rod with a pestleshaped end, which is introduced at E. The distribution vessel Vis then detached at F and is closed (under a hydrogen blanket) by a ground glass plug. A connection is then made at H to a flexible, corrugated tombac tube, which leads to the pump and the hydrogen generator. Rubber tubing is not suitable for this purpose because it gives off moisture. Finally, the powder is distributed into the thin-wall glass bulbs K and tubes M by tilting vessel V. The bulbs and the tubes are melt-sealed. The samples in K are used for analysis; therefore the bulbs are weighed together with their ground joints before assembly and after sealing off. The material thus obtained is very pure.



Fig. 257. Preparation of magnesium phosphide and magnesium arsenide.

II. A porcelain boat  $10 \times 15 \times 60$  mm.—a boat made of  $Al_2O_3$  or MgO is recommended for the metal—is filled with very fine Mg shavings turned from a solid metal block. A second boat is filled with 6 g. of purified red P (or 14 g. of sublimed As). Both boats are pushed to the closed end of a 500-mm. combustion tube which has a diameter just sufficient to accommodate the boats. The tube is closed with a rubber stopper fitted with a stopcock and evacuated

by means of an oil pump while being gently fanned with a flame. Even better, the tube is drawn out using a torch, and a section 350 mm. long is sealed off under high vacuum.

After the metal has been heated to dull red heat, the P (or As) is heated with a second burner and distilled onto The reaction of the two elethe Mg. ments is accompanied by bright incandescence. In order to avoid removal of P (or As) from the reaction site while heating the Mg or during the reaction itself, the center portion of the reaction tube is heated by a short multiple-tube burner, so that little or no P or As condenses on the cold surface of the rubber stopper. This also assures that the stopper will be able to relieve any pressure buildup in the tube.

In order to remove excess P (or As) from the finished product, the tube is shifted so that only the two boats are heated by the multiple-tube burner. Heating is continued for 30 minutes. In the case of the arsenide, the excess



Fig. 258. Bottling of magnesium phosphide under CO<sub>2</sub> blanket.

nonmetal can sometimes be removed only after pulverizing the reaction product. After cooling, the tube is filled with dry  $CI_2$ . The boat is taken out and the moisture-sensitive substance is sealed into a prepared ampoule as shown in Fig. 258.

#### PROPERTIES:

The phosphide is bright yellow, the arsenide is brown-red. Stable in completely dry air at room temperature, decompose in moist air ( $Mg_3P_2$  decomposes faster than  $Mg_3As_2$ ). Both crystallize in structure type  $D5_g$  (carbon sesquioxide). d ( $Mg_3P_2$ ) 2.055; ( $Mg_3As_2$ ) 3.148.

#### **REFERENCES:**

- I. H. Gautier, Comptes Rendus Hebd. Séances Acad. Sci. <u>128</u>, 1168 (1899); E. Zintl and E. Husemann, Z. phys. Chem. (B) <u>21</u>, 138 (1933).
- II. F. Weibke, Thesis, Technische Hochschule, Hannover, 1930; P. Ehrlich, unpublished studies.

# Magnesium Carbides

# MgC<sub>2</sub>, Mg<sub>2</sub>C<sub>3</sub>

These compounds cannot be prepared from carbon and the metal since  $MgC_2$  decomposes below  $500^{\circ}C$  and  $Mg_2C_3$  above  $700^{\circ}C$ . However, relatively pure products are obtained by passing gaseous hydrocarbons over heated MgO powder (60 microns or smaller). Prior to use, this extremely fine powder is activated by heating under vacuum for a short period of time.

I a. To prepare  $MgC_2$ , about 8 g. of Mg powder is placed in a 10-cm. -long iron boat and covered with some steel wool. The boat is then inserted into a porcelain tube of about 4 cm. diameter. The air is displaced by repeated evacuation and filling with H<sub>2</sub>. The tube is then heated at  $700^{\circ}$ C for 40 minutes while a slow stream of  $H_2$  is passing through it and cooled to  $450^{\circ}C$  at 11 mm. over a period of 10 minutes. Finally, acetylene is introduced over a period of 15 minutes, until atmospheric pressure is restored. Passage of  $C_{2}H_{2}$  is then continued for one hour at a flow rate of 6 liters/hour and a temperature of  $450^{\circ}$ C. The tube is then removed from the furnace and cooled in a slow H<sub>2</sub> stream. The steel-blue to black, very hard product is ground at once under absolutely dry ether and stored. The crude carbides are purified by dissolving the excess Mg with ethyl bromide in ether, which may be done in a small Soxhlet extraction apparatus. A product containing 70% MgC<sub>2</sub> is obtained.

I b. The preparation of  $Mg_2C_3$  is similar. As above, the air is displaced from the reaction tube, which is then heated to 850°C for one hour. After lowering the temperature, pentane is passed at 700°C for two hours and 710°C for one hour, under slightly reduced pressure (aspirator suction). The pentane is introduced into the reaction tube by means of a dropping funnel. A gas flow rate of 2.5 liters/hour corresponds to 35 ml., or 30 drops per minute. Cooling in a H<sub>2</sub> stream is carried out as above. The light to dark-gray products contain up to 85% Mg<sub>2</sub>C<sub>3</sub>.

Other preparative methods: II. Reaction of  $MgCl_2$  with  $CaC_2$  (does not, however, result in a purer product). III. Reaction of a ether solution of diethylmagnesium and acetylene.

PROPERTIES:

MgC<sub>2</sub> has a tetragonal and Mg<sub>2</sub>C<sub>3</sub> a hexagonal structure. MgC<sub>2</sub> decomposes at about 550°C, forming Mg<sub>2</sub>C<sub>3</sub> and C; Mg<sub>2</sub>C<sub>3</sub> decomposes at 740 to 750°C into its elements. On reaction with water, MgC<sub>2</sub> and Mg<sub>2</sub>C<sub>3</sub> give C<sub>2</sub>H<sub>2</sub> and C<sub>3</sub>H<sub>4</sub>, respectively.

REFERENCES:

I. F. Irmann, Helv. Chim. Acta 31, 1584 (1948).

- II. A. Schneider and J. F. Cordes, Z. anorg. allg. Chem. <u>279</u>, 94 (1955).
- III. W. H. Rueggeberg, J. Amer. Chem. Soc. 65, 602 (1943).

### **Magnesium Silicide**

### Mg<sub>2</sub>Si

I.

 $\begin{array}{rrrr} 2 \ Mg \ + \ Si \ = \ Mg_2Si \\ 48.6 \ \ 28.1 \ \ 76.7 \end{array}$ 

An intimate blend of Mg filings and pulverized Si (3:1) is charged into a MgO boat and heated under high vacuum. The reaction begins at 450°C, lasts only a few minutes, and is accompanied by a large volume increase. The product contains an excess of free Mg, but no free Si. The metal can either be distilled off at higher temperature (700°C) or extracted by treating the pulverized alloy with ethyl iodide in the presence of anhydrous ether, or with an ether solution of bromobenzene containing a grain of iodine. After washing with ether, the product is first dried at room temperature, then at 300°C. The slate-blue, shiny crystals have the composition Mg<sub>2</sub>Si.

II.

 $SiO_2 + 4Mg = Mg_2Si + 2MgO$ 60.1 97.3 76.7 80.7

When larger amounts of silicide are required for the synthesis of silanes, the compound is best prepared as follows:

Precipitated silicic acid, free of P and S and containing 0.3 to 0.5% (based on the weight of the anhydrous material) alkaline residue after evaporation is dehydrated by heating at bright red heat for several hours. The carefully pulverized anhydride is intimately blended with twice its amount of Mg powder. This mixture (100 g.) is ignited in an iron crucible of about 1000 ml. capacity, which is well cooled by a large quantity of cold water; the reaction rapidly propagates throughout the entire mass, generating incandescent white heat. Immediately after the onset of the reaction, the crucible is covered with a lid equipped with a gas inlet tube, and a fast  $H_2$  stream is passed over the reactants. Some Mg is forced outside and burns on the lid. After cooling, the product cake adhering to the bottom of the lid can be easily removed.

Other preparative methods: III. Claims have been advanced that high purity  $Mg_2Si$  can be prepared by melt electrolysis of magnesium silicate.

PROPERTIES:

Rather hard, very brittle, slate-blue crystals. Crystallizes in structure type C1 (fluorite type). d 1.94. Stable to alkalis. Decomposed by acids, forming silicon hydrides and hydrogen.

**REFERENCES:** 

- I. G. Gire, Comptes Rendus Hebd. Séances Acad. Sci. <u>196</u>, 1405 (1933); P. Lebeau and P. Bossuet, Comptes Rendus Hebd. Séances Acad. Sci. <u>146</u>, 284 (1908); L. Wöhler and O. Schliephake, Z. anorg. allg. Chem. <u>151</u>, 1 (1926).
- II. A. Stock and C. Somieski, Ber. dtsch. chem. Ges. 49, 115 (1916). III. J.-L. Andrieux, Congr. Chim. Ind. Nancy 18, I, 124 (1938).

# Magnesium Germanide

## Mg<sub>2</sub>Ge

 $2 Mg + Ge = Mg_2Ge$ 48.6 72.6 121.2

A homogeneous mixture of finely powdered Ge and Mg (3:2) is introduced into a Pyrex tube. The latter is heated by means of a Bunsen burner, while a flow of H<sub>2</sub> is passed through. The temperature is raised gradually. As red heat is approached, an incandescent reaction sets in at one spot and then spreads through the whole mass without further heating.

PROPERTIES:

Dark gray granular product. Characteristic odor of  $\text{GeH}_4$  due to reaction with air moisture. M.p. 1115°C.

REFERENCES:

- L. M. Dennis, R. B. Corey and R. W. Moore, J. Amer. Chem. Soc. 46, 657 (1924).
- W. Klemm and H. Westlinning, Z. anorg. allg. Chem. 245, 365 (1940).

# Calcium, Strontium, Barium Metals

The available methods of preparation include: I) fusion electrolysis; II) aluminothermic reaction; III) decomposition of azides. The first method (used exclusively in industry) has only occasional laboratory application. Relevant literature references for Ca are listed under I. Method II does not give good yields with Ca, but is applicable to Sr and Ba. Method III does not yield pure metal and may be used only with small quantities of material, since explosions are possible. The large surface area of the finely divided metal obtained by this method may be useful for special purposes. The procedures for Sr and Ba are similar to those given for Ca.

## CALCIUM, Ca

In most cases commercial metal, purified by distillation, is used as the starting material.

*Purification by distillation*: a) Ca is distilled in an iron tube by directing metal vapor against a steel, nickel or copper cooling finger, which is polished at the lower end. A flange connects the pipe to a high vacuum system. Even

pipe to a high vacuum system. Even better is an apparatus made of a quartz or porcelain tube r, sealed at one end and provided with a ground joint. A tall iron tubular crucible tis inserted into the tube. The latter, reaching into the cold zone, provides protection against corrosion. The arrangement and dimensions are shown in Fig. 259.

For ease of disassembly the high vacuum connection should be made directly at the quartz tube r, rather than at the cap. The tube should thus be elongated accordingly. In such an arrangement the connection to high vacuum need not be broken while the distillate is being removed.

Crude Ca (40 g.) is placed in a crucible (1-mm.-thick walls) made of electrolytic iron, stainless steel or low-carbon steel. To prevent contamination of the distillate with a fine dust consisting of residual calcium oxide and calcium nitride, the metal is covered with a thin layer of steel wool. The latter is well degreased before use and ignited in moist  $H_{2}$ , as is the crucible, in order to effect-



Fig. 259. Purification of calcium by distillation.

ively remove P and C. The crucible must be separated from the round bottom of the quartz tube by a narrow porcelain piece p, so that it will not burst the tube on a sharp temperature change. A water-cooled, thin-wall finger f runs axially through a 10-cm.-long glass tube passing through the ground cap. The rod is cemented

to the tube with picein, forming an airtight connection. The metal parts must be perfectly smooth and free of oxide. The ground joint is cooled by means of a lead coil carrying water. The apparatus may be arranged vertically so that the cold finger will be suspended without stress in the guide tube of the ground cap. The same apparatus can, however, be operated more conveniently when tilted at about  $30^{\circ}$ . Aside from this the ground joints are easier to cool in this case.

Before it is put into operation, the apparatus should be tested for leaks under high vacuum. A good diffusion pump (with a capacity of 15-22 liters/sec.) should be in continuous operation in order to maintain a sufficiently high vacuum, since large volumes of gas contained in the crude material are evolved during distillation (especially in the initial stage). The pump is best connected to the apparatus by way of a trap cooled with liquid  $N_2$ . The tubing that connects the pump with the distilling apparatus should be as short and as large diameter as possible.

During the first phase of distillation, at  $700^{\circ}$ C, alkali metals (primarily Na, together with some Ca) are deposited upon the cold finger (Mg cannot be separated from calcium by distillation). The apparatus is allowed to cool somewhat and is then filled with purified Ar (or with dry CO<sub>2</sub> if completely cool). At the same time the cold finger and the ground cap are replaced by fresh ones. The alkali metals occasionally ignite on contact with the air when the apparatus is opened.

The main distillation step is carried out under high vacuum  $(\leq 10^{-3} \text{ mm.})$  at the lowest possible temperature, so as to ensure high-purity metal. The last fraction is discarded. At 850°C the calcium deposit on the cold finger builds up as grape clusters of long silver-white, luminous crystallites, which will not tarnish to any appreciable extent on brief exposure to air. At higher temperatures the distillates obtained are richer in chlorine.

The structural characteristics of the separated metal depend markedly on the distillation rate. If the operating temperature is raised by 100-150°, the metal is deposited more rapidly and will be more compact. The temperature of the cold finger is likewise of importance. The lower the temperature, the smaller the particle size of the deposited metal. With air cooling of the finger the metal will separate in the form of rhombohedra.

After the thoroughly cooled apparatus has been filled with Ar or  $CO_2$ , the tube is opened and the metal dislodged from the cold finger with a spatula or, if necessary, with suitable tongs or forceps. This is best done under toluene or in a cylinder filled with  $CO_2$ .

It is advisable to repeat the distillation several times. The Ca obtained by this procedure is 99.7-99.9% pure and contains a few hundredths of one percent of O, N, Cl, Fe, Si, Mg. Preparation of high-purity calcium by preliminary distillation followed by fractional distillation (2 mm. He) is described by W. J. McCreary, J. Metals <u>10</u>, 615 (1958).

b) The procedure is simplified for lower purity metal. Use of the cold finger is omitted, as is the interruption of distillation for separation of alkali metals.

Crude Ca (20 g.) is placed in a tubular crucible made of lowcarbon iron (inside diameter 20 mm., length 150 mm.) smoothly machined on the outside. An open iron pipe (length 150 mm.) smoothly machined on the inside, jackets the crucible with a clearance of only a few tenths of a millimeter. An iron wire is welded to the jacketing tube for more convenient handling. Both the crucible and the jacket must be well cleaned mechanically before the experiment and reduced in moist hydrogen.

The crucible and protective pipe are inserted into a quartz tube tilted about  $30^{\circ}$  from the horizontal. A ground cap connects the tube with a high-vacuum system. The distillation proceeds under the same conditions as described above. The iron jacket reaches up to the cooling coil. The temperature gradient along this cap is such that the alkali metals are deposited on its upper portion (which often results in ignition on opening of the reactor), while the Ca condenses in the form of beads somewhat below the height corresponding to the rim of the furnace. One of the disadvantages of this arrangement is that the condensation temperature of the calcium is sufficiently high so that the product reacts with the iron of the tube and thus becomes somewhat contaminated.

After cooling, the Ca is chipped off the wall with a chisel. This is best done when the pipe is cut open. If it is desired to compact the metal, the first crucible is removed and a shorter one substituted. This crucible has a larger diameter than the jacket, which dips into it. The Ca can then be melted down from the jacket in the same quartz reactor simply by shifting the furnace. It is best to melt under argon at atmospheric pressure, since too large quantities of the metal vaporize under vacuum.

c) A mild steel apparatus with a capacity of 200 g. of metal is used for distilling larger amounts of Ca.

Remelting: If compact Ca rather than a sponge is desired, the latter must be remelted. The pure distilled Ca (20-25 g.) is ground under Ar or CO<sub>2</sub> with an iron pestle in a crucible (diameter 23 mm., height 75 mm., wall thickness 1 mm.) made of electrolytic iron or low-carbon steel and preignited in moist H<sub>2</sub>. A second crucible is fitted snugly into the first, pushed as far down as possible and welded to it at the rim level. The lower part of the outer crucible is cooled with water. The Ca melts at 900°C after brief heating. On cooling, the crucible is cut open and the compact Ca cylinder is easily loosened from the wall. The metal does not segregate, but the metal surface adhering to the rim absorbs a few hundredths of a percent of Fe, which may be removed by turning the cylinder on a lathe.

PROPERTIES:

Atomic weight 40.08. Silver-white metal. M.p. 850°C, b.p. 1439°C, d 1.55. The m.p. is lowered considerably by nitrides and other impurities.

Calcium is as soft as Pb. The cubic face-centered  $\alpha$ -Ca transforms at 464°C into the hexagonal  $\gamma$ -Ca.

The  $\beta$ -Ca (between 300 and 464°C), until now considered a separate modification, is actually an alloy of Ca and impurities. The purer the metal, the more slowly it tarnishes in the air. It reacts rather slowly with water at ordinary temperatures, but the reaction becomes more vigorous on warming. With dilute acids, the reaction is violent.

**REFERENCES:** 

General: H. Funk, Die Darstellung der Metalle im Laboratorium [Laboratory Preparation of Metals], Stuttgart, 1938, p. 29; H. Grubitsch, Anorganisch-präparative Chemie [Inorganic Preparative Chemistry], Vienna, 1950, p. 396; G. Chaudron in: A. E. van Arkel, Reine Metalle [Pure Metals], Berlin, 1939, p. 126; I. F. Smith, O. N. Carlson and R. W. Vest, J. Electrochem. Soc. <u>103</u>, 409 (1956).

I. *Electrolysis*: O. Ruff and W. Plato, Ber. dtsch. chem. Ges. <u>35</u>, 3612 (1903); W. Muthmann, H. Hofer and L. Weiss, Liebigs Ann. Chem. <u>320</u>, 231 (1902); B. Neumann and E. Bergve, Z. Elektrochem. <u>20</u>, 187 (1914); M. Trautz, ibid. <u>21</u>, 130 (1915).

Purification by distillation: a) F. W. Dafert and R. Miklauz, Mh. Chem. <u>34</u>, 1685 (1913); W. Biltz and G. F. Hüttig, Z. anorg. allg. Chem. <u>114</u>, 242 (1920); W. Biltz and W. Wagner, ibid. <u>134</u>, 1 (1924); P. S. Danner, J. Amer. Chem. Soc. <u>46</u>, 2382 (1924); A. von Antropoff and E. Germann, Z. phys. Chem. <u>137</u>, 209 (1928); P. Remy-Genneté, Ann. Chim. (10) <u>19</u>, 263 (1933); W. C. Johnson et al., J. Amer. Chem. Soc. <u>61</u>, 318 (1939); P. Ehrlich, unpublished experiments. b) Private communication from Prof. Dr. W. Fischer, Hannover. c) W. D. Treadwell and J. Sticher, Helv. Chim. Acta 36 1822 (1953).

Remelting: O. Ruff and H. Hartmann, Z. anorg. allg. Chem. <u>121</u>, 167 (1922); W. Biltz and W. Wagner, ibid. <u>134</u>, 1 (1924).

STRONTIUM, Sr

I. Strontium can be prepared by fusion electrolysis (see references for calcium, part I), by the aluminothermic procedure (II), and by decomposition of azide (III). Strontium prepared by the aluminothermic process, as well as the commercially available metal, is purified by distillation under high vacuum.

п.

$$3 \operatorname{SrO} + 2 \operatorname{Al} = 3 \operatorname{Sr} + \operatorname{Al}_2 \operatorname{O}_3$$
  
310.9 53.9 262.9 101.9

Reaction II is endothermic. It goes to completion because the alkaline earth metals are highly volatile, and thus the equilibrium is continually and favorably shifted by the use of high vacuum.

The apparatus described above (Ca distillation) is charged with a homogeneous mixture of 60 g. of SrO, freshly ignited at  $1100^{\circ}$ C, and 14 g. of pure Al shot. Larger quantities of the reactants should not be used since the vigorous reaction will cause unnecessarily heavy losses through spattering. The mixture is preheated for an hour. The heating is then continued for four hours at  $1010-1030^{\circ}$ C under high vacuum ( $10^{-2}$  mm.). The reduction to powder and the mixing should be done rapidly, so as to keep the oxide as free as possible from hydroxide and carbonate. In spite of these precautions, a temporary deterioration of the vacuum occurs during the heating from 500 to  $800^{\circ}$ C.

The yield of 98% pure metal is 20-30 g. Strontium reacts rapidly with atmospheric moisture. Consequently, all operations such as opening the apparatus, Sr transfer, etc., should be done in an atmosphere of dry  $CO_2$  or, better, under purified Ar.

The arrangement for distillation of commercial Sr is also the same as in the Ca procedure. Forty grams of crude Sr can be distilled in four hours at 1030°C. Repeated distillation yields 99.9% pure metal.

ш.

$$\frac{\mathrm{Sr}(\mathrm{N}_{\mathbf{3}})_2}{171.7} = \frac{\mathrm{Sr}}{87.6} + \frac{3}{84.1} \mathrm{N}_2$$

This method is the same as described for Ba under III. It yields a finely divided black metal powder, which is strongly contaminated with nitride (>10%) and ignites immediately on exposure to air.

PROPERTIES:

Silver-white metal, softer than Ca. M.p. 757°C, b.p. 1364°C; d 2.6. Crystallizes in structure type A1.

Tarnishes in air (becoming yellow-brown) and is finally coated by a layer of white oxide. Finely divided Sr ignites on exposure to air.

**REFERENCES:** 

II. Aluminothermic reaction: A. Guntz et al., Ann. Chim. Phys. [8] 10, 437 (1907); Comptes Rendus Hebd. Séances Acad. Sci. 143, 339 (1906); 151, 813 (1910); Bull. Soc. Chim. France [4] 35, 712 (1924).

General: See also references listed for Ca.

## BARIUM, Ba

I. The procedures are the same as for Ca and Sr. The pure metal is obtained by repeated redistillation under high vacuum.

II.  $3 \text{ BaO} + 2 \text{ Al} = 3 \text{ Ba} + \text{Al}_{2}\text{O}_{3}$ 460.1 53.9 412.1 101.9

The directions for preparing this metal are the same as for Sr (II), with minor modifications: 72 g. of BaO, nearly free of peroxide, and 11 g. of aluminum shot are heated for one hour. The heating is continued for five more hours at  $1100^{\circ}$ C under high vacuum ( $10^{-2}$  mm.). The yield is < 20 g. of 97-98% pure metal. Commercial BaO usually contains some Sr, which concentrates in the distilled metal. Of the three alkaline earth metals, Ba is the most reactive. Every precaution must be taken when opening the apparatus or transferring the metal, since the deposited metal is finely crystalline and easily ignites in the presence of traces of moisture, especially while being detached from the cold finger. The safest way is to carry out this step under toluene.

In purifying Ba by distillation, the directions given for Ca should again be followed. Forty grams of crude Ba may be distilled at  $1050^{\circ}$ C in four hours. Triple distillation yields 99.6% pure metal.

III. 
$$Ba(N_s)_2 = Ba + 3 N_2$$
  
221.4 137.4 84.0

Small quantities of Sr and Ba can be obtained by decomposition of the corresponding azides under vacuum. The metals are obtained as finely divided, highly reactive black powders. They may be used in various reactions, which can be carried out directly in the equipment used for preparation. Their isolation, i.e., removal from the apparatus, is hardly possible, because on exposure to air these metals react immediately with ignition. Another problem is contamination of the metal with nitride (>10%) due to a side reaction of the type: Ba + 2Ba(N<sub>3</sub>)<sub>2</sub> = Ba<sub>3</sub>N<sub>2</sub> + 5N<sub>2</sub>. By subjecting the Ba(N<sub>3</sub>)<sub>2</sub> to rapid decomposition, the nitride content can be kept down to a low level.

The decomposition equipment consists of a distillation flask (250 ml.) which is connected to a manometer and a vacuum pump. To avoid excessive loss of metal by entrainment, a piece of glass wool is inserted into the outlet tube.

The flask is charged with 10 g. of the azide and sealed on top. After evacuating the flask, heat (small flame) is applied at one point. Occasionally an explosion occurs, accompanied by fire, and the flask of  $Ba(N_3)_2$  then becomes coated with a black metal film.

928

The decomposition begins at  $160^{\circ}$ C; that of Sr(N<sub>3</sub>)<sub>2</sub> at  $140^{\circ}$ C. Once started, it can be carried to completion at a lower temperature (120°C for Ba and 110°C for Sr).

PROPERTIES:

Silver-white metal. M.p. 710°C, b.p. 1537°C; d 3.76. Crystallizes in structure type A2.

On exposure to air Ba quickly turns gray and finally black. Ignites very readily and reacts very vigorously with water.

**REFERENCES:** 

III. Azide decomposition: E. Tiede, Ber. dtsch. chem. Ges. <u>40</u>, 1742 (1916); P. Remy-Genneté, Ann. Chim. [10] <u>19</u>, 263 (1933). General: See references given for Ca and Sr.

## **Calcium Strontium and Barium Hydrides**

## CaH<sub>2</sub>, SrH<sub>2</sub>, BaH<sub>2</sub>

Ca (Sr, Ba) +  $H_2 = CaH_2$  (Sr $H_2$ , Ba $H_2$ ) 40.1 87.6 137.4 22,4 l 42.1 89.6 139.4

Following repeated redistillation under vacuum, the metal is milled free of the adhering oxide under argon. When the surface of the metal is clean and bright, it is placed in the hydrogenation apparatus described for the preparation of alkali hydrides (p. 971). Air must be excluded. Thoroughly purified and dried electrolytic  $H_2$  is employed for the hydrogenation. The metal is placed in a boat made of pure electrolytic iron, which is inserted into a quartz tube connected to a manometer. A thin-wall tubular insert made of electrolytic iron is fitted into the quartz tube to protect its heated portion from chemical attack.

Hydride formation usually begins between 400 and 500°C for Ca and Sr and between 200 and 300°C for Ba. Thereafter the temperature is raised to 1000°C. As soon as the absorption of  $H_2$  is completed, the reactor is slowly cooled. The hydride is now ready for use, provided air and moisture are absent.

The finely crystalline substances thus obtained resemble the mineral serpentine and retain the fibrous structure of the original metals. The interior of the CaH<sub>2</sub> product in most cases contains a residue consisting of unreacted Ca metal. As the temperature rises above  $1000^{\circ}$ C the compound, in a stream of H<sub>2</sub>, becomes overheated and dissociation begins. The metal evaporates and the components recombine in the colder areas of the tube. However,

slow distillation yields the hydrides as colorless, lustrous crystals, about 1 mm. wide. A residue is left in the boat.

To arrive directly at very pure hydrides, a special apparatus is used. This is provided with a mechanical arrangement for continuous grinding of the reaction product as it is formed during the hydrogenation. W. D. Treadwell and J. Sicher, Helv. Chim. Acta 36, 1938 (1953), used such equipment to obtain nearly 99.9% pure  $\overline{CaH_2}$ .

ANALYSIS:

The hydride samples are removed from the apparatus in the absence of air and moisture and are then decomposed with air-free water under vacuum. The  $H_2$  formed in the decomposition is cooled to a low temperature to condense out the moisture. It is then dried over  $P_2O_5$ , transferred to a gas burette by means of a Toepler pump, and measured.

PROPERTIES:

Colorless, lustrous rhombic crystals, the stability decreasing from  $CaH_2$  to  $BaH_2$ . The hydrides react vigorously with water, evolving  $H_2$ . The heats of formation from the elements are remarkably high.

d (CaH<sub>2</sub>) 1.90; (SrH<sub>2</sub>) 3.27; (BaH<sub>2</sub>) 4.15.

**REFERENCES:** 

E. Zintl and A. Harder, Z. Elektrochem. 41, 33 (1935).

Simplified procedures yield substances of lower purity (~90%):

C. B. Hurd and K. E. Walker, J. Amer. Chem. Soc. <u>53</u>, 1681 (1931).

P. Remy-Genneté, Ann. Chimie (10) 19, 263, 353 (1933).

W. C. Johnson, M. F. Stubbs, A. E. Sidwell and A. Pechukas, J. Amer. Chem. Soc. <u>61</u>, 318 (1939).

# Calcium, Strontium, Barium Halides

The anhydrous halides of Ca, Sr and Ba are prepared in the same way as described for the corresponding compounds of Be and Mg.

Anhydrous iodides can also be obtained from the hydrides by using nonaqueous solvents as reaction media. In pyridine solution, for example, the reaction is:  $BaH_2 + 2NH_4I = BaI_2 + 2NH_3 + 2H_2$ . Most of the pyridine is then distilled off and the residual, bound solvent is removed under vacuum at 150-160°C. The yield is 97%.

REFERENCE:

M. D. Taylor and L. R. Grant, J. Amer. Chem. Soc. 77, 1507 (1955).

# Calcium Oxide CaO

I.

 $CaCO_3 = CaO + CO_2$ 100.1 56.1 44.0

Calcium oxide is obtained by igniting calcium carbonate or calcium oxalate at about 800°C.

*Purification of calcium salts:* Very pure calcium oxide, such as needed in the preparation of phosphorus, is obtained according to Tiede and Riemer in the following way:

Carrara marble is dissolved in very pure nitric acid. The  $CO_2$  is completely removed by boiling and the hot solution is treated with  $Ca(OH)_2$  solution to precipitate traces of salts of foreign metals (primarily Cu, Fe and Mg). The filtrate is heated almost to boiling and  $CO_2$  is passed through. The bicarbonate is formed in proportion to the amount of added  $Ca(OH)_2$  and is subsequently decomposed by boiling the solution until it is neutral. The Ca is finally precipitated as the carbonate, carrying along traces of Fe. The filtered  $Ca(NO_3)_2$  solution is treated with a concentrated solution of pure  $(NH_4)_2CO_3$  which contains one third by volume of concentrated ammonia. The  $CaCO_3$  precipitates on cooling. The precipitate is washed well, dried and ignited to the oxide. A quartz crucible and an electric furnace are used for this last reaction. For best results, no more than 2-3 g. should be prepared at a time.

Other methods: II. A purification procedure including an even greater number of steps is described in a thesis by Riemer. The process yields  $CaCO_3$  and CaO of extreme purity such as employed in the production of luminescent materials. III. Very pure  $CaCO_3$ is prepared according to the method of Richards and Honigschmid. Calcium nitrate solution is slightly acidified with nitric acid and treated with a slight excess of pure  $Ca(OH)_2$  solution to precipitate Fe(OH)<sub>3</sub> and most of the Mg(OH)<sub>2</sub>. Impurities consisting of Ba, Sr and Mg salts are removed by repeated recrystallization of the nitrate. The  $(NH_4)_2CO_3$  required for the precipitation of the carbonate is purified by distillation with water. Further details will be found in the original report.

#### PROPERTIES:

Crystallizes in structure type B1. M.p. 2572°C, b.p. 2850°C; d 3.35. Treatment with water yields the hydroxide.

#### REFERENCES:

- I. E. Tiede: Private communication, 1949.
- II. F. Riemer: Thesis, Berlin, 1920.

III. T. W. Richards and O. Hönigschmid, J. Amer. Chem. Soc. <u>32</u>, 1577 (1910); E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 71.

# Strontium Oxide

## SrO

I. Strontium oxide can be obtained by heating pure  $SrCO_3$  in a stream of  $H_2$  at 1300°C for several hours, by dehydration of  $Sr(NO_3)_2 \cdot 4H_2O$  with subsequent calcining at 1100°C for 1.5 hours, or by dehydration of  $Sr(OH)_2$  above 850°C. If a high-purity product is desired the heating (as in the case of BaO) is done in a stream of  $N_2$  or  $H_2$ , free of  $O_2$  and  $CO_2$ . The vessels must be made of nickel or sintered alumina.

Purification of strontium salts: II. Highest purity SrO is obtained by way of the nitrate, which is prepared as follows. Strontium carbonate is treated with 500 ml. of distilled water in a fiveliter flask. This suspension is slowly dissolved in 830-840 ml. of concentrated nitric acid. Next, 7 ml. of concentrated  $H_2SO_4$  is added to precipitate most of the Ba, and the solution is brought to boiling. The hot solution (pH~6) is treated for 30 minutes with  $H_2S$ , previously passed through Ba(OH)<sub>2</sub> solution. After the precipitate has settled, it is suction-filtered through a glass frit to avoid contamination of the filtrate with dust.

The filtrate is treated with 45 ml. of saturated ammonium oxalate solution and the Ca precipitated on addition of ammonia to a pH of 7. The mixture is brought to boiling and, while hot, treated again with  $H_2S$ . It is left to stand overnight and filtered the next morning. Nitric acid is added until the pH is reduced to 3, and 2-3 ml. of  $Br_2$  is added to oxidize Fe and Mn. Excess  $Br_2$  is removed by boiling. The solution is then made alkaline with ammonia (pH 8) and  $H_2S$  is admitted briefly. After standing a few hours the solution is filtered and the filtrate is reacidified with nitric acid (pH 3). After heating to boiling, the filtration is repeated. The  $Sr(NO_3)_2$  solution is by this time sufficiently free of heavy metal impurities.

III. The last traces of Ca and Mg can be removed by precipitating the Sr as  $SrSO_4$ . In most cases, however, repeated recrystallization (3 or 4 times) of strontium nitrate in Pt or quartz vessels will suffice. By passing very pure  $NH_3$  and  $CO_2$ , Sr can be precipitated as  $SrCO_3$  from a solution containing 200 g. of  $Sr(NO_3)_2 \cdot 4H_2O$  in one liter of water. The precipitate is washed 8 to 10 times by decantation. After filtration and drying in an electric furnace it is calcined to form the oxide. PROPERTIES:

Formula weight 103.63. White powder. M.p. 2430°C; d 5.02. Crystallizes in structure type B1.

Converted to Sr(OH)<sub>2</sub> by moisture or water.

REFERENCES:

- I. A. Guntz and F. Benoit, Bull. Soc. Chim. France [4] <u>35</u>, 712 (1924); G. F. Hüttig and A. Arbes, Z. anorg. allg. Chem. <u>192</u>, 225 (1930).
- II. A. L. Smith, R. D. Rosenstein and R. Ward, J. Amer. Chem. Soc. <u>69</u>, 1725 (1947).
- III. T. W. Richards, Proc. Amer. Acad. <u>30</u>, 376 (1894); E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 75.

### **Barium Oxide**

#### BaO

I. Thermal decomposition of the nitrate (which melts and decomposes with effervescence), iodate or peroxide is used to obtain the oxide. The final temperature must be above  $800^{\circ}$ C. Normally the compound is heated in air to  $1100^{\circ}$ C, to remove any peroxide impurity present.

The highest purity product is obtained by thermal decomposition of BaCO<sub>3</sub> under high vacuum. The most expedient procedure is to liberate most of the CO<sub>2</sub> at 950°C and the remainder at 1100-1150°C. This prevents melting of the product.

Another method for preparing BaO is based on careful dehydration of  $Ba(OH)_2 \cdot 8H_2O$ , which has been repeatedly recrystallized before use. The mass is then heated for two hours at  $800^{\circ}C$  in a stream of dry N<sub>2</sub> or H<sub>2</sub>, free of O<sub>2</sub> and CO<sub>2</sub>. Melting of the Ba(OH)<sub>2</sub> cannot be avoided since the temperature reaches  $700^{\circ}C$ ; hence, it is advisable to use vessels made of sintered corundum or pure nickel. Other materials such as Pt, stainless steel, quartz, porcelain, etc., are corroded.

Purification of barium salts: II. Impurities which can be separated as sulfides are removed by following the directions given for SrO (II). The hydroxide octahydrate is not altogether suitable for further purification (removal of residual traces of Sr, Ca, Mg, etc.) since its recrystallization, while removing all traces of Ca, does not lead to complete removal of Sr. The nitrate, on the other hand (according to Richards), is very well suited for removal of the last traces of Sr, Ca, Mg, K and Na from Ba salts. Very pure commercial barium nitrate is recrystallized eight times from very pure water in Pt containers. If Pt is not available, quartz vessels may be used. The mother liquor is separated by centrifuging in a Pt tube, this procedure being about ten times as efficient as suction filtration. At the beginning, as well as toward the end of the recrystallization process, the nitrate solution is filtered through a fine glass frit. Every precautionary measure to exclude dust and harmful vapors is observed during the procedure. The simplest way to achieve this is to use the particular laboratory premises for no other work but the above procedure.

To obtain  $BaCO_3$  (as the starting material for various Ba salts), a hot, pure barium nitrate solution is precipitated, with  $(NH_4)_2CO_3$ , which is prepurified by distilling its aqueous solution through a Pt condenser into a Pt receiver. The  $BaCO_3$  is separated from the mother liquor in a Pt centrifuge tube.

PROPERTIES:

Formula weight 153.36. White powder. M.p. 1923<sup>o</sup>C;d 5.98. Crystallizes in structure type B1.

Converted to the hydroxide by moisture. The carbonate is formed on exposure to the  $CO_2$  of the air.

**REFERENCES:** 

I. G. F. Hüttig and A. Arbes, Z. anorg. allg. Chem. <u>196</u>, 403 (1931).

II. O. Hönigschmid and R. Sachtleben, Z. anorg. allg. Chem. <u>178</u>, 1 (1929); E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 77.

## **Calcium Hydroxide**

 $Ca(OH)_2$ 

Calcium hydroxide is formed on addition of water to CaO, provided the latter has not been overheated during calcination. Another way is to treat aqueous solutions of calcium salts with alkalies.

$$\begin{array}{rl} \text{Ca(NO_3)}_2 \cdot 4 \text{ H}_2\text{O} \ + \ 2 \text{ KOH} \ = \ \text{Ca(OH)}_2 \ + \ 2 \text{ KNO}_3 \ + \ 4 \text{ H}_2\text{O} \\ 236.2 & 112.2 & 74.1 & 202.2 & 72.1 \end{array}$$

I. Boiled water (500 ml.) is used to dissolve 46 g. of Ca(NO<sub>3</sub>)<sub>2</sub>. 4H<sub>2</sub>O. To this solution, 500 ml. of 1N C. P. potassium hydroxide solution (CO<sub>2</sub> free) is added in small portions with shaking, the temperature being kept at 0°C. The product is washed a number of times with a total of 12 liters of water, filtered and again washed on the filter. The precipitate is dried for 20 hours under vacuum over sulfuric acid (d 1.355); it then has a composition corresponding to its formula.

In all of these procedures the  $CO_2$  of the air must be carefully excluded.

II. Crystalline Ca(OH)<sub>2</sub> can be obtained by the diffusion method. Two 50-ml. beakers are placed in a vessel equipped with a removable lid. One of them contains 30 g. of recrystallized CaCl<sub>2</sub>.  $6H_2O$  dissolved in 50 ml. of  $H_2O$ ; the other contains 12 g. of NaOH in 50 ml. of  $H_2O$  and a small quantity of Ba(OH)<sub>2</sub> to precipitate the carbonate. Enough water is poured into the vessel to cover the beakers 2 cm. above the rim. After four weeks, the 1-cm.-long crystals are collected on a filter crucible and washed quickly with water, dilute hydrochloric acid, water, alcohol and ether. They are then dried for a short time at  $110^{\circ}C$ .

III. Small crystals can be obtained in a few hours by treating a  $Ca(OH)_2$  solution, saturated at  $10^{\circ}C$ , with 20 ml. of 20% potassium hydroxide solution.

PROPERTIES:

Hexagonal crystals, which decompose before melting. At a water vapor pressure of 10 mm., the decomposition temperature is  $380^{\circ}$ C. Moderately soluble in water. d 2.08.

## **REFERENCES:**

I. G. F. Hüttig and A. Arbes, Z. anorg. allg. Chem. <u>191</u>, 161 (1930).
 II. J. Johnston and C. Grove, J. Amer. Chem. Soc. <u>53</u>, 3976 (1931).
 III. C. Nogareda, Anales Soc. Espanola Fisica Quim. 29, 556 (1931).

# Strontium Hydroxide

## $Sr(OH)_2 \cdot 8 H_2O$ , $SrO \cdot 9 H_2O$

When SrO is moistened with the theoretical amount of water, a vigorous reaction occurs, with formation of  $Sr(OH)_2$  (whitepowder) and evolution of heat. On further addition of water, the mono-, hepta-, or octahydrate is formed. Because of its low solubility, the octahydrate can be prepared from any soluble Sr salt by precipitating with a strong base.

 $\begin{array}{c} \mathrm{Sr}(\mathrm{NO}_3)_2 \cdot 4 \mathrm{H}_2\mathrm{O} + 2 \mathrm{KOH} + 4 \mathrm{H}_2\mathrm{O} = \mathrm{Sr}(\mathrm{OH})_2 \cdot 8 \mathrm{H}_2\mathrm{O} + 2 \mathrm{KNO}_3 \\ 283.7 & 112.2 & 72.1 & 265.8 & 202.2 \end{array}$ 

A solution containing 20 g. of  $Sr(NO_3)_2 \cdot 4H_2O$  in 40 ml. of distilled water is cooled to  $0^{\circ}C$  and treated in the absence of  $CO_2$  with the equivalent amount of KOH dissolved in 100 ml. of  $H_2O$ . The alkali is added drop by drop. The precipitate is washed with 1.5 liters of  $H_2O$  and is then free of both K and  $NO_3$  ions.

PROPERTIES:

Prismatic, tetragonal crystals. The first mole of water of crystallization apparently is more readily given off than the others. The dehydration curve shows that, at p = 10 mm. H<sub>2</sub>O, Sr(OH)<sub>2</sub> is stable from 100 to  $450^{\circ}$ C. M.p. 375°C.

**REFERENCE:** 

G. F. Hüttig and A. Arbes, Z. anorg. allg. Chem. 192, 225 (1930).

## Calcium, Strontium, Barium Peroxides

CaO<sub>2</sub>, SrO<sub>2</sub>, BaO<sub>2</sub>

 $CaO_2 \cdot 8 H_2O$ ,  $SrO_2 \cdot 8 H_2O$ ,  $BaO_2 \cdot 8 H_2O$ 

### CaO<sub>2</sub>

In common with the corresponding magnesium compound, and in contrast to  $SrO_2$  and  $BaO_2$ ,  $CaO_2$  cannot be obtained through direct reaction of  $O_2$  with either CaO or Ca. It is found only by heating  $CaO_2 \cdot 8H_2O$  above 130°C. Nearly anhydrous peroxide,  $CaO_2 \cdot 0.38$  H<sub>2</sub>O, is obtained by direct precipitation from aqueous solutions, e.g., by treating 11 g. of  $CaCl_2 \cdot 6H_2O$ , dissolved in 50 ml. of 3% H<sub>2</sub>O<sub>2</sub>, with 7 ml. of 25% ammonia in 100 ml. of H<sub>2</sub>O. The temperature should either be above  $60^{\circ}C$ , or the amount of water should be reduced to 30 ml. at  $20^{\circ}C$ .

### SrO<sub>2</sub>

This compound is usually prepared by slow (several hours) dehydration of the octahydrate at  $300^{\circ}$ C. It can also be obtained, but not entirely water-free (0.68 mole of H<sub>g</sub>O), by precipitating a solution of 5 g. of Sr(NO<sub>3</sub>)<sub>2</sub> in 5 ml. of 30% H<sub>2</sub>O with 7 ml. of 25% ammonia at 55°C.

Very pure  $SrO_2$  can be prepared from SrO under an  $O_2$  pressure of 200 to 250 atm. at 350-400°C. The starting material must be free from hydroxide or carbonate and the  $O_2$  should be thoroughly dried.

936

## BaO2.

The peroxide is prepared by careful dehydration of the octahydrate, first in a desiccator under reduced pressure, and then in a drying pistol over  $P_2O_5$  at  $100^{\circ}C$ .

According to Bernal et al., preparations containing 100% BaO<sub>2</sub> (or SrO<sub>2</sub>) may be obtained only by drying for one month in O<sub>2</sub> at room temperature.

High grade  $BaO_2$  can also be prepared by heating loose BaO at  $500^{\circ}C$  in a stream of  $O_2$  which has been thoroughly dried and freed of  $CO_2$ .

### PROPERTIES:

Formula weights:  $CaO_2$  72.08;  $SrO_2$  119.63;  $BaO_2$  169.36. Of the three peroxides (all white)  $BaO_2$  is the most stable and its solubility in water is the highest. When immersed in water, the peroxides gradually form the octahydrates at room temperature. Decomposed by acids with liberation of  $H_2O_2$ . Both  $SrO_2$  and  $BaO_2$ crystallize in the tetragonal system. They are face-centered and are isostructural with  $CaC_2$ .

**REFERENCES:** 

- CaO<sub>2</sub>: R. de Forcrand, Comptes Rendus Hebd. Séances Acad. Sci. <u>130</u>, 1250, 1308, 1388 (1900); E. H. Riesenfeld and W. Nottebohm, Z. anorg. allg. Chem. 89, 408 (1914).
- SrO<sub>2</sub>: E. H. Riesenfeld and W. Nottebohm, Z. anorg. allg. Chem. <u>89</u>, 408 (1914); C. Holtermann and P. Lafitte, Comptes Rendus Hebd. Seances Acad. Sci. <u>208</u>, 517 (1939).
- BaO<sub>2</sub>: P. Askenasy and R. Rose, Z. anorg. allg. Chem. <u>189</u>, 1 (1930); J. D. Bernal, E. Djatlowa, I. Kasarnowski, S. Reichstein and A. G. Ward, Z. Kristallogr. <u>92</u>, 344 (1935); C. Engler and W. Becker, Ber. Heidelberger Akad. No. 15, 5 (1909/10).

## THE OCTAHYDRATES

The peroxide hydrates are formed when alkaline solutions of alkaline earth salts are treated with  $H_2O_2$ . To avoid the formation of dihydrates or of anhydrous salts, the following directions must be complied with when working with a Ca salt, for example: 11 g. of  $CaCl_2 \cdot 6H_2O$  is dissolved in 5 ml. of  $H_2O$  and treated with 50 ml. of 3%  $H_2O_2$ . To this solution, 7 ml. of 25% ammonia in 100 ml. of  $H_2O$  is added.

The procedure is the same as for the preparation of  $SrO_2 \cdot 8H_2O$ . At room temperature  $BaO_2 \cdot 8H_2O$  is formed only in strongly alkaline solutions. An excess of  $H_2O_2$  must be avoided. For example, 100 ml. of Ba(OH)<sub>2</sub> solution, previously saturated at  $14^{\circ}$ C, is treated with 5 ml. of 3% H<sub>2</sub>O<sub>2</sub>.

PROPERTIES:

Formula weights:  $CaO_2 \cdot 8H_2O$  216.20;  $SrO_2 \cdot 8H_2O$  263.75;  $BaO_2 \cdot 8H_2O$  313.48.

Lustrous, white crystals. The three compounds are isomorphous. In air, they become opaque and are slowly converted to the carbonates by the  $CO_2$ . They hydrolyze in water and are dehydrated in absolute alcohol.

REFERENCES:

- E. H. Riesenfeld and W. Nottebohm, Z. anorg. allg. Chem. <u>89</u>, 405 (1914).
- C. Nogareda, Anales Soc. Espanola Fisica Quim. 28, 475 (1930).

## Calcium, Strontium, Barium Sulfides

CaS, SrS, BaS

 $CaCO_3$  (SrCO<sub>3</sub>, BaCO<sub>3</sub>) + H<sub>2</sub>S = CaS (SrS, BaS) + H<sub>2</sub>O + CO<sub>2</sub> 100.1 (147.6, 197.3) 34.1 72.2 (119.7, 169.4) 18.0 44.0

Alkaline earth sulfides can be easily prepared in small quantities (3-5 g.) by heating their pure carbonates (C.P.) for about two hours at about 1000°C in a fast stream of an equimolar mixture of  $H_2S$  and  $H_2$ . When water ceases to evolve,  $H_2$  alone is passed through for about half an hour to decompose the polysulfides. The product is left to cool in a stream of  $H_2$ .

Other preparative methods: Larger amounts of sulfides, though of lower purity, can be obtained by heating the carbonates in a crucible with an excess of elemental S. Here, tight closure of the crucible is essential and use of an autoclave is advantageous. This procedure is mostly used to produce phosphors based on alkaline earth sulfides.

PROPERTIES:

White powders (BaS is often grayish). M.p. >  $2000^{\circ}$ C.d CaS 2.59; SrS 3.65; BaS 4.36. Crystallize in structure type B1. Oxidize in dry air and are decomposed by moisture and, more rapidly, by acids, with which they evolve H<sub>2</sub>S.

938

**REFERENCES:** 

- P. Sabatier, Comptes Rendus Hebd. Séances Acad. Sci. 88, 651 (1879); Ann. Chim. Phys. 5, 22, 6 (1881).
- E. Tiede: Private communication; see also H. von Wartenberg, Z. anorg. allg. Chem. 252, 136 (1943) and A. Guntz and F. Benoit, Bull. Soc. Chim. France (4) 35, 712 (1924).

## Calcium, Strontium, Barium Selenides

#### CaSe, SrSe, BaSe

I. CaSeO<sub>4</sub>, (SrSeO<sub>4</sub>, BaSeO<sub>4</sub>) + 4  $H_2$  = CaSe (SrSe, BaSe) + 4  $H_2O_{183.0}$  (230.6 280.3) 8.1 119.0 166.6 216.3 72.1

Small quantities of alkaline earth selenides are obtained by the method of Berzelius, through reduction of the corresponding selenates in a  $H_2$  stream. The water formed in this reaction decomposes the selenides, forming  $H_2$ Se. In turn, the latter is thermally cleaved into  $H_2$  and Se, imparting a reddish hue to the preparation. It is therefore highly important to work with a fast  $H_2$  stream and use only small amounts of starting material.

The selenates are prepared from the corresponding alkaline earth nitrates. The salt is added to a concentrated solution of  $K_2SeO_4$  (H<sub>2</sub>SeO<sub>4</sub> neutralized with potassium hydroxide). The precipitate is filtered off and dried at 200°C.

The selenate (about 1 g.) is distributed in a thin layer over a quartz boat 10 cm. long and 1 cm. wide. The drying at 200°C is repeated, this time in a reaction tube and in a stream of  $N_2$ . Reduction in a stream of  $H_2$  follows. It should continue for two hours between 400-500°C for CaSeO<sub>4</sub>, at 600°C for SrSeO<sub>4</sub>, and at 500°C for BaSeO<sub>4</sub>. A pure white product is obtained for SrSe and BaSe, but in the case of CaSe the white color occasionally shows a reddish tint.

To avoid undesirable decomposition caused by air moisture, transfer of the product from the boat must be carried out in the absence of air. Special devices are used for this purpose. These are connected to the reaction tube by means of ground-glass joints, so that the boat contents can be emptied into a side attachment (see Part I, p. 75).

II. 
$$SrSeO_3 + 2 NH_3 = SrSe + 3 H_2O + N_2$$
$$214.6 \quad 34.1 \quad 166.6 \quad 54.1 \quad 28.0$$

Pure SrSe is best obtained from the selenite by high-temperature reduction with  $NH_3$ . The SrSeO<sub>3</sub> is prepared by dissolving repeatedly sublimed SeO<sub>2</sub> in water and adding the theoretical amount

of hot strontium nitrate solution (for a special method of purifying the salt, see Smith, Rosenstein and Ward). After neutralizing the solution with ammonia, the strontium selenite is filtered off, washed six times with water until free of nitrate, and dried at  $200^{\circ}$ C. The salt is reduced in a fast stream of NH<sub>3</sub> at  $860^{\circ}$ C. The reaction is completed in 2-3 hours.

When  $NH_3$  is used as the reducing agent, side reactions take place.

Occasionally,  $N_4Se_4$  forms in the colder portion of the tube and explodes when the boat is taken out.

The same method is used to prepare CaSe.

PROPERTIES:

White powders. d CaSe 3.82; SrSe 4.54; BaSe 5.02. Crystallize in structure type B1.

In air, the powders acquire a reddish color within a few minutes and turn light brown in a few hours. Decomposed by water. Treatment with HCl produces  $H_2Se$  gas, and red Se separates.

**REFERENCES:** 

- I. F. A. Henglein and R. Roth, Z. anorg. allg. Chem. <u>126</u>, 227 (1923); E. Tiede and E. Blasius, unpublished experiments.
- II. A. L. Smith, R. D. Rosenstein and R. Ward, J. Amer. Chem. Soc. 69, 1725 (1947).

Calcium, strontium and barium tellurides, although not perfectly pure, can be obtained similarly by reduction of the tellurates in a hydrogen stream (CaTe at  $680^{\circ}$ C, SrTe at  $690^{\circ}$ C, BaTe at  $580^{\circ}$ C).

REFERENCE:

M. Haase, Z. Kristallogr. 65, 509 (1927).

## Calcium, Strontium, Barium Nitrides

Ca<sub>3</sub>N<sub>2</sub>, Sr<sub>3</sub>N<sub>2</sub>, Ba<sub>3</sub>N<sub>2</sub>

Ca<sub>3</sub>N<sub>2</sub>

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\begin{array}{rrrr} 3\,Ca\,+\,N_{2}\,=\,Ca_{3}N_{2}\\ 120.3\,&28.0\,&148.3 \end{array}
```

Distilled, finely divided Ca metal is placed in a Ni boat and in a nitrogen stream for 3-4 hours at  $450^{\circ}$ C. At this temperature the

nitriding is particularly rapid, because the lattice of the metal becomes less compact as a result of structural changes in the crystals (transition point).

Contradictory data have been published on the nitriding temperature for Ca. The reaction has been reported to proceed at a measurable rate only above  $800^{\circ}$ C. The discrepancies, it seems, can be traced to use of metal which is not quite pure, or of N<sub>2</sub> still containing traces of O<sub>2</sub>.

Small amounts of Na vapor absorbed by the surface of the metal prevent the formation of a continuous nitride film; hence, they activate the metal. The latter, in the active form, is an agent for purifying argon; see p. 82.

When Ca metal is treated with  $NH_3$  at 800°C, some hydride is formed simultaneously.

### **PROPERTIES:**

 $a-Ca_{3}N_{2}$  (structure type D5<sub>3</sub>) transforms into  $\beta-Ca_{3}N_{2}$  at 700°C. Depending on the temperature of formation the nitride powder is colored black (350°C) to golden yellow (1150°C), or else acquires mixed coloration (at intermediate temperatures). d 2.62. Decomposes in water to Ca(OH)<sub>2</sub> and NH<sub>3</sub>.

### REFERENCES:

- F. W. Dafert and R. Miklauz, Monatsh. Chem. 34, 1685 (1913).
- P. Dutoit and A. Schnorf, Comptes Rendus Hebd. Seances Acad. Sci. 187, 300 (1928).
- A. von Antropoff and E. German, Z. phys. Chem. 137, 209 (1928).
- J. Rieber, Thesis, Technische Hochschule, Hannover, 1930.
- M. von Stackelberg and R. Paulus, Z. phys. Chem. B 22, 305 (1933).

### Sr<sub>3</sub>N<sub>2</sub> and Ba<sub>3</sub>N<sub>2</sub>

The nitriding temperatures for Sr and Ba are 460 and 560°C, respectively. To achieve complete conversion to the nitride, the heating must be continued for a long time or the temperature must be raised to  $700-750^{\circ}$ C on cessation of absorption of N<sub>2</sub>.

#### REFERENCES:

F. W. Dafert and R. Miklauz, Monatsh. Chem. 34, 1685 (1913). A. Guntz and F. Benoit, Ann. Chim. (9) 20, 15 (1923).

# **Barium Azide**

## Ba(N<sub>3</sub>)<sub>2</sub>

 $2 \operatorname{NaN}_{3} + 2 \operatorname{H}_{2}\operatorname{SO}_{4} = 2 \operatorname{NaHSO}_{4} + 2 \operatorname{HN}_{3};$ 130,0 196,2 240,2 86,0  $2 \operatorname{HN}_{3} + \operatorname{Ba}(\operatorname{OH})_{2} = \operatorname{Ba}(\operatorname{N}_{3})_{2} + 2 \operatorname{H}_{2}\operatorname{O}_{86,0}$ 86.0 171.4 221.4 36.0

Hydrazoic acid is formed on dropwise addition of sulfuric acid (1:1) to NaN<sub>3</sub> solution. The acid is next distilled into a receiver, which contains a Ba(OH)<sub>2</sub> suspension (2/3 of the calculated amount). The distillation temperature should be about  $60^{\circ}$ C (or lower, if the pressure is reduced). A few drops of phenolphthalein are added to the reaction mixture and, toward the end of distillation, the remaining Ba (OH) is added continually to a neutral reaction (disappearance of red color). The work must be done under a good hood (strong draft), because the HN<sub>3</sub> vapor is highly toxic.

Another way of preparing the azide is to place the entire quantity of Ba(OH) in the receiver and use a higher dilution. The excess of hydroxide is then removed by passage of  $CO_2$ , and the precipitated BaCO<sub>3</sub> is filtered off.

The Ba( $N_3$ )<sub>2</sub> solution is crystallized in a vacuum desiccator over CaCl<sub>2</sub>. The resultant monohydrate is completely dehydrated over  $P_2O_5$ . A preliminary recrystallization of the salt is advisable.

The strontium salt is prepared by a similar procedure.

PROPERTIES:

Crystallizes in cylindrical crystals. On impact, explodes with mild force, bursting into flames. d 2.94.

REFERENCE:

P. Remy-Genneté: Ann. Chimie (10) 19, 289 (1933).

# Calcium Phosphide

 $Ca_3P_2$ 

I.

To prepare this salt, distilled Ca is heated with red P in a thoroughly evacuated combustion tube. The heating is continued until the reaction becomes spontaneous. This is accompanied by incandescence. Milder reaction conditions result in a higher quality product. To this end, Ca is heated to dull red heat in a corundum boat, and P is slowly distilled onto the Ca. After cooling, the sealed end of the tube is broken off, and the tube is filled with dry  $CO_a$ , using the same procedure as described for Mg<sub>3</sub> P<sub>2</sub> (II).

To ensure that no excess metal remains in the preparation, the product must be heated again with P for a long time at  $600^{\circ}$ C. An evacuated, sealed tube is used for this purpose.

Where purity requirements are particularly high, Ar should be used as carrier gas, following the directions given for  $Mg_3P_2$  (I).

II. 
$$3 \operatorname{Ca}_{3}(\operatorname{PO}_{4})_{2} + 16 \operatorname{Al} = 3 \operatorname{Ca}_{3}\operatorname{P}_{2} + 8 \operatorname{Al}_{2}\operatorname{O}_{3}$$
  
93.1 43.2 54.7 81.6

If  $Ca_3P_2$  is to be used only as a raw material for the preparation of  $PH_3$ , it can be prepared by the aluminothermic method. The drawback of this procedure is that the product phosphide cannot be separated from the  $Al_2O_3$ .

Powdered, predried calcium phosphate (232 g.) and 108 g. of Al shot are ground to a homogeneous mixture. An igniting mixture is used to kindle the reaction. In most cases it is necessary to preheat the crucible to  $500^{\circ}$ C.

## PROPERTIES:

Crystalline, red-brown powder. d 2.51. Decomposes slowly in moist air and more vigorously in water yielding  $Ca(OH)_2$  and  $PH_3$ .

### **REFERENCES:**

- I. F. Weibke, Thesis, Technische Hochschule, Hannover, 1930.
- II. C. Matignon and R. Trannoy, Comptes Rendus Hebd. Séances Acad. Sci. <u>148</u>, 167 (1909).

# **Calcium Carbide**

# CaC<sub>2</sub>

The CaC<sub>2</sub> obtained on reversal of the reaction producing CaCN<sub>2</sub> or on heating a mixture of CaCN<sub>2</sub> and C under high vacuum is much purer than that from the synthesis from the elements or the reaction between CaO and C.

$$CaCN_2 + C = CaC_2 + N_2$$
  
80.1 12.0 64.1 28.0

I.

 $2 \operatorname{CaCN}_{2} = \operatorname{CaC}_{2} + 2 \operatorname{N}_{2} + \operatorname{Ca}_{160.2} \quad 64.1 \quad 56.0 \quad 40.1$ 

A sintered clay boat is charged with pure  $CaCN_2$ , either alone or with an added amount (somewhat below the calculated value) of well-charred sugar charcoal or acetylene black. The boat is inserted into a ceramic tube (inside diameter 30 mm., length 500 mm.). The tube is provided with a ground stopper at one end and sealed at the other. It is connected to a high-vacuum system and heated by means of an electric furnace with a molybdenum wire winding to temperatures above  $1350^{\circ}C$ .

The cyanamide (or the mixture with carbon) is heated in two stages. The first heating is continued for 2-3 hours at 1100-1150°C to remove most of N<sub>2</sub>, which is drawn off under high vacuum. It is imperative that the temperature does not exceed 1170°C, since this is the eutectic temperature for the system  $CaCN_2-CaC_2$ . By that time the cyanamide is so far decomposed that mixed crystals of  $CaC_2$  and  $CaCN_2$  cannot be formed during the second heating stage at a higher temperature (1350°C, one hour), and thus the last traces of N<sub>2</sub> are quantitatively expelled. The product is pure white and contains over 99%  $CaC_2$ , with no impurities, except traces of CaO and C.

The very slight corrosion of the  $Al_2O_3$  boat by the CaCN<sub>2</sub> cannot be entirely avoided. However, the loosely adhering product layer is easily detached.

 $CaO + 3C = CaC_2 + CO$ 56.1 36.0 64.1 28.0

Laboratory preparation of calcium carbide via the reaction of pure CaO with very pure carbon in an electric arc at  $2000^{\circ}$ C is carried out as follows.

A large porcelain crucible, at least 80 mm. in diameter on top and 60 mm. high, is filled with a well-blended, dry mixture of equal parts of quicklime and wood charcoal to a level 10 cm. below the rim. The ingredients are not too finely powdered to prevent elutriation losses during later gas evolution. The crucible is placed on top of a brick. Two carbon rods (carbon welding electrodes, or rods made of electrolytic graphite), at least 15 mm. in diameter and 200 mm. long, are tapered to a point at their lower ends, while slotted (and thus flexible) brass caps are affixed on their upper ends. Each cap has a clamp screw, which serves as an electric terminal. A horizontal hole is drilled through each carbon rod and cap combination, and a 1-2 mm. connective copper wire is fitted snugly into the hole and bent back at both ends so that the cap is securely attached to the electrode. The electrodes are attached to a stand in such a way that they reach

II.

down to the center of the crucible, and their points are about 10 mm. apart. The asbestos insulated clamps connecting the electrodes to the stand are attached just below the brass caps. The lime-charcoal mixture is piled up in the center of the crucible and the latter is then covered with an asbestos sheet. Insulated copper wires (cross section 16 mm.) connect the electrodes to the power supply. The electrodes are in series with 0-50 amp. ammeter, a 40 amp. rheostat (6 ohms at 220 v., 3 ohms with a line voltage of 110 v.) and a double-pole knife switch. Where a suitable rectifier is available the use of direct current is preferred since a D.C. arc is far smoother than an A.C. arc. The potential across the electrodes is measured with a voltmeter.

The current is switched on with the rheostat set at maximum. It takes some time before the electric arc is initiated. The current is then set at 30-40 amp. The voltmeter should register a potential of 50-70 v. If the reading is much higher, the carbons are too far apart (and vice versa). The current must be shut off before any adjustment is made. When the operation is properly conducted, long tongues of burning CO escape from the crucible together with occasional puffs of dust from the charge. The current is shut off after 5-10 min. and the crucible is left to cool. A few grams of sintered or lump calcium carbide will be found under the electrode ends.

Other methods: III. Heating distilled Ca metal with C produces dark-colored or black carbides which are 94% pure, at best. IV. Technical grade carbide, 75-89% pure, can be enriched to 92% grade (containing residual Ca and C) by remelting a number of times at  $2100^{\circ}$ C, separation of strata, and compressing.

### PROPERTIES:

When pure, colorless and crystalline. M.p.  $\sim 2300$  C; d 2.22. The structure of pure CaC<sub>2</sub> differs from that of the commonly used "technical" carbide which is contaminated with impurities (pseudo-cubic, tetragonal, face-centered). Decomposes in water, evolving acetylene.

REFERENCES:

- I. H. H. Franck, M. A. Bredig and K. H. Kou, Z. anorg. allg. Chem. <u>282</u>, 75 (1937).
- II. W. Fischer, private communication; see also H. Grubitsch, Präparative Anorganische Chemie, Vienna, 1950, p. 333.
  II. E. Botolfsen, Ann. Chim. [9] 18, 5 (1922); O. Ruff and E.
- III. E. Botolfsen, Ann. Chim. [9] 18, 5 (1922); O. Ruff and E. Foerster, Z. anorg. allg. Chem. 131, 321 (1923); O. Ruff and B. Josephy, Z. anorg. allg. Chem. 153, 17 (1926); H. H. Franck, M. A. Bredig and K. H. Kou, Z. anorg. allg. Chem. 232, 75

(1937); see also H. H. Franck and H. Endler, Z. phys. Chem. (A) 184, 127 (1939).

IV. O. Ruff and E. Foerster, Z. anorg. allg. Chem. 131, 321 (1923).

## Calcium Cyanamide

### CaCN<sub>2</sub>

 $\begin{array}{rl} CaCO_{3} + 2 HCN = CaCN_{2} + CO + H_{2} + CO_{2} \\ 100,1 & 54,0 & 80,1 & 28,0 & 2,0 & 44,0 \end{array}$ 

Pure CN, previously dried over  $CaCl_2$  and  $P_2O_5$ , is condensed in a receiver cooled to a low temperature. The amount used is three times the stoichiometric quantity. A stream of N<sub>2</sub> is bubbled, preferably mixed with NH<sub>3</sub>, through the receiver and becomes laden with HCN. It is then passed over  $CaCO_3$ , which fills a porcelain boat inserted into a porcelain tube. The cooled HCN receiver is warmed to  $18^{\circ}$ C when a reaction temperature of 700 to  $850^{\circ}$ C is reached in the porcelain tube, but not before. Heating for three hours yields aperfectly white, 99.4% pure product (34.8% N).

PROPERTIES:

Colorless crystals. M.p.  $\sim 1200^{\circ}$ C. Gradually decomposed by water. Simultaneous treatment with CO<sub>2</sub> and water liberates free cyanamide. Warming a cyanamide solution to  $70^{\circ}$ C yields urea.

**REFERENCE:** 

H. H. Franck and H. Heimann, Angew. Chem. 44, 372 (1931).

## **Calcium Silicides**

CaSi, CaSi<sub>2</sub>

CaSi

$$\begin{array}{c} Ca + Si = CaSi \\ 40.0 \quad 28.1 \quad 68.1 \end{array}$$

A mixture of Ca chips and pure Si (15% excess) is placed in a boat made of unglazed hard porcelain. The boat is immediately pushed into the hot  $(1000^{\circ}\text{C})$  zone of a quartz tube through which a CO<sub>2</sub> stream is passed. Within a few seconds a vigorous reaction

sets in and the mass begins to melt. The boat is then withdrawn from the hot zone, thus immediately quenching the product. When crushed to a powder, the grayish-black, porous, solid mass disintegrates into CaSi flakes, which have a metallic luster. The thin crust of CaO is easily removed.

### PROPERTIES:

Covering CaSi with dilute hydrochloric acid causes vigorous decomposition. Spontaneously igniting silanes are formed, with white silicic acid as the residue.

CaSi<sub>2</sub>

I.

 $\begin{array}{r} 3\,{\rm CaO}\,+\,5\,{\rm Si}\,=\,2\,{\rm CaSi}_2\,+\,{\rm CaSiO}_3\\ 168.2\,\,\,140.3\,\,\,192.4\,\,\,\,116.1 \end{array}$ 

Following Goldschmidt's procedure,  $CaSi_2$  is prepared by fusing high purity CaO (53.6%) and Si (26.4%) in the presence of suitable fluxes (12%  $CaF_2$  and 8%  $CaCl_2$ ) at 1400°C. The melt must be thoroughly stirred with an  $Al_2O_3$  rod so that the molten  $CaSi_2$  will separate on the surface of the melt. The hot, viscous reaction mixture disintegrates on cooling, freeing silicide particles. The product obtained by this method is always rich in Si.

Larger charges result in improved yields.

II. Simple fusion of the elements does not produce pure CaSi<sub>2</sub>. Therefore, for smaller quantities and higher purity, it is preferable to heat CaSi with the stoichiometric amount of Si.

CaSi	+ Si =	CaSi <sub>2</sub>
68.1	28.1	96.2

The mixture is placed in a Ni boat and heated in a stream of  $H_2$  at 1000°C. The last phase of conversion proceeds very slowly and requires up to 15 hours of heating.

Other methods: CaSi<sub>2</sub> can also be prepared, according to Dodero, by melt electrolysis above  $1000^{\circ}$ C using a flux. The proportions of the components are:  $3SiO_2 + 3CaCO_3 + 6CaF_2 + CaCl_2$ .

### PROPERTIES:

Hexagonal lead-gray tablets, with a bright metallic luster. M.p.  $1020^{\circ}$ C; d 2.5. Crystallizes in structure type C12. If not sufficiently cooled with ice, reaction with HCl is violent and the disilicide dissolves while a characteristic greenish-yellow product separates out.

For the preparation of strontium and barium silicides, see Wöhler and Schuff. The reaction proceeds as in the case of calcium silicide, through at a somewhat higher temperature.
REFERENCES:

- L. Wöhler and F. Müller, Z. anorg. allg. Chem. 120, 49 (1922).
- L. Wöhler and W. Schuff, Z. anorg. allg. Chem. 209, 33 (1932).
- M. Dodero, Comptes Rendus Hebd. Séances Acad. Sci. 198, 1593 (1934).

## Calcium Germanide

### CaGe

I.

Ca + Ge = CaGe40.0 72.6 112.6

Calcium turnings are ground to a fine powder in a ball mill under anhydrous benzene. Traces of benzene are then removed under vacuum.

Using the method described on p. 712, GeO<sub>2</sub> is reduced to metal powder with the aid of H<sub>2</sub>.

The metal powders are mixed in stoichiometric proportions in an alundum boat. The boat is placed in an evacuated quartz tube and the reaction is started by bringing the boat contents to red heat. Suddenly a bright glow appears at one spot, spreading within seconds throughout the boat and causing partial pulverization of its The reaction is finished at that point, but the product contents. still contains some unreacted Ge. This happens even when an excess of calcium is used.

22.41

II.  $CaH_{*} + Ge = CaGe + H_{*}$ 42.0 72.6 112.6

Germanium powder is mixed with an equimolar quantity of CaH<sub>2</sub>, previously pulverized under N<sub>2</sub> blanket. The hydride is taken in slight excess. The mixture is placed in an iron boat and the latter inserted into an electrically heated ceramic tube. The atmosphere within the tube is inert at that point. The tube is then connected to a high-vacuum system. The reaction begins at about 450°C and is finished at about 950°C, provided the H<sub>2</sub> formed in the process is removed at regular intervals. The temperature is then raised to 1000°C and gas removal continued for another half hour to achieve complete removal of the last traces of H<sub>2</sub>. After cooling under vacuum, the tube is opened and the product is pulverized as rapidly as possible and placed in sealed ampoules. It still contains traces of Ca.

**PROPERTIES:** 

Dark-gray powder. Rapidly turns yellow on exposure to atmospheric moisture.

REFERENCES:

P. Royen and R. Schwarz, Z. anorg. allg. Chem. <u>211</u>, 412 (1933).
 With regard to Ca<sub>2</sub>Ge see also P. Eckerlin and E. Wölfel, ibid. <u>280</u>, 321 (1955).

# SECTION 18

Alkali Metals E. DÖNGES

# Alkali Metal Compounds from Minerals

### LITHIUM CARBONATE FROM LEPIDOLITE

Lepidolite contains 1.2-6% (average 4-5%) LiO<sub>2</sub>. According to a patent (Metallbank, Metallurgische Ges. A.G.), lepidolite is converted to Li<sub>2</sub>CO<sub>3</sub> (and LiF or Li<sub>3</sub>PO<sub>4</sub>) in the following way.

Lepidolite is decomposed with concentrated  $H_2SO_4$  and is then well calcined to decompose aluminum and iron sulfates. The aqueous extract of the roasted material contains impurities composed of salts of Al, Fe, Mg, Mn, etc. Most of these are separated by treatment with lime and calcium hypochlorite (to oxidize Mn). This purified liquor contains, in addition to Li and K sulfates (e.g., 6.6 g. of lithium/liter), traces of other salts such as CaSO<sub>4</sub> and occasionally MnSO<sub>4</sub> and MgSO<sub>4</sub>. For this reason, a concentrated solution (about 1%) of Li<sub>2</sub>CO<sub>3</sub>, LiF, or Li<sub>3</sub>PO<sub>4</sub> is added to the liquor until a precipitate ceases to form. About 100–150 ml. of the carbonate solution is required per liter of liquor, depending on how well the latter was prepurified. It is better to precipitate the last traces of impurities with a lithium salt, rather than with some other alkali salt, because an excess of the former precipitating agent causes no loss of original Li.

The liquor is then repeatedly purified until the  $\text{Li}_2\text{CO}_3$ , precipitated on addition of pure  $K_2\text{CO}_3$  and filtered, washed and dried, is of analytical priority. Since the solubility of  $\text{LiCO}_3$  is very high, large quantities of Li are still present in the mother liquor. This solute can be precipitated as the phosphate or the fluoride. These salts will also be very pure. Needless to say, all the Li can be precipitated as pure phosphate or fluoride without going through the partial precipitation of the carbonate.

This method can also be used for the purification of technical grade Li salts contaminated with Ca, Mg, Fe, etc.

Solubility of  $\text{Li}_2\text{CO}_3$  (0°C) 1.54; (20°C) 1.33; (100°C) 0.73 g./100 g. of  $\text{H}_2\text{O}$ .

Solubility of LiF (18°C) 0.27 g./100 g. of  $H_2O$ .

Solubility of  $\text{Li}_3\text{PO}_4$  (25°C) 1 g./3360 g. of H<sub>2</sub>O.

REFERENCES:

Metallbank, Metallurgische Gesellschaft A. G., German Patent 413723 (1925). Inventor H. Weidmann.

## RUBIDIUM AND CESIUM CHLORIDES FROM CARNALLITE

Natural carnallites contain, on the average, 0.02% RbCl and 0.0002% CsCl. "Synthetic carnallites," which are obtained as intermediates in the manufacture of potassium chloride through recrystallization of natural carnallite, have a much higher Rb and Cs content.

I. According to Jander and Faber, as well as Jander and Busch, the first step in the simultaneous production of RbCl and CsCl from synthetic carnallite is recrystallization of the latter. Thus, synthetic carnallite (6.5 g.) is boiled in 2.3 liters of water. The hot liquor (d. 1.3) is suction filtered on a Büchner funnel to separate the solid KCl precipitate. The residue is placed in a dish and treated with 400 ml. of hot water to extract all soluble material. Next. it is washed on the filter with 150 ml. of cold water. While still somewhat moist it weighs about 1300 g. and is free from both Rb and Cs. A sample dissolved in dilute HCl and treated with a silicomolybdate solution (see below) forms no precipitate even after standing for several hours.] The combined filtrates are evaporated until crystallization begins. The "second synthetic carnallite" precipitated overnight from the cooled solution amounts to about 2270 g. and contains all the Rb and Cs. The concentration of these elements is thus three times higher than in the first synthetic carnallite. The mother liquor gives no precipitate on addition of a silicomolybdate solution even if allowed to stand for many hours.

The second synthetic carnallite can be further enriched by repetition of the crystallization. However, this is not necessary.

To precipitate the silicomolybdate salts, 2.3 kg. of the second carnallite is dissolved in 2.9 liters of warm water and treated with 0.9 liter of concentrated hydrochloric acid. The KCl (about 70 g., free of Rb and Cs) which separates out on cooling to room temperature is filtered off and the solution is reheated to  $60-70^{\circ}$ C. Vigorous agitation is then started and the solution is treated with sufficient amount of molybdosilicate solution (see p. 953) to precipitate one-tenth of the total available silicomolybdate Rb<sub>4</sub>[SiMo<sub>12</sub>O<sub>40</sub> · 2H<sub>2</sub>O] and Cs<sub>4</sub>[SiMo<sub>12</sub>O<sub>40</sub> · 2H<sub>2</sub>O]. The required quantity of solution is determined on a sample. Usually, more reagent will be needed than indicated by stoichiometry. The precipitate starts to form on cooling to 40-50°C and requires 12-15 hours for complete settling.

The clear supernatant liquid is then decanted. The scaly precipitate, which adheres to the walls of the container, detaches after brief drying in air and is quantitatively collected. The precipitate now contains all of the Cs and one tenth of the Rb. The Cs:Rb ratio is now 1:10, instead of the original 1:100. The silicomolybdate precipitate is then worked up to obtain the CsCl. It is placed in a porcelain boat, which is then inserted in a Pyrex tube. If larger quantities are handled, the precipitate may also be placed in a second tube and distributed in a layer not more than 0.5 cm. This second tube is then concentrically inserted into the thick. first. The assembly is heated in an electric furnace to 450°C while HCl gas, saturated with CCl, in a wash bottle filled with liquid CCl<sub>4</sub>, is allowed to flow through the tube. The CCl<sub>4</sub> vapor is thermally decomposed to C<sub>2</sub>Cl<sub>6</sub> and Cl<sub>2</sub>. The latter is to prevent the formation of a small amount of volatile low-valence Mo compounds. Loose crystalline deposits of Mo(OH) 2Cl<sub>2</sub> are formed in the rear, cold portion of the tube. The exit HCl gas is passed through a water-filled wash bottle to absorb any entrained Mo(OH) <sub>2</sub>Cl<sub>2</sub>. To remove all traces of Mo from the product mixture of alkali chlorides. Cl<sub>2</sub> is passed through the apparatus for a short time. This is done toward the end of the reaction, which lasts 1.5 hours. During the final stage of conversion the temperature is increased to incipient red heat.

The pure white residue consists of RbCl, KCl, CsCl and SiO<sub>2</sub>. It is repeatedly extracted with hot water, and the silicic acid is filtered off. The filtrate is evaporated to dryness. To separate KCl and RbCl from CsCl, 16 g. of dry residue is dissolved in 40 ml. of warm 2.5N HCl and treated with 50 ml. of warm 96% alcohol. After cooling, the first KCl-RbCl precipitate is filtered off and the filtrate is heated and again treated with 300 ml. of warm 96% alcohol. Upon cooling, the second RbCl-KCl precipitate is filtered off and the filtrate once more treated with 50 ml. of alcohol and filtered as above. The filtrate is evaporated; the residue contains all the CsCl. It is dissolved in 20 ml. of 2.5N HCl and treated with 5 ml. of 20% SbCl<sub>3</sub> solution in 7.5N HCl. After a while, the Cs precipitates out as the chloroantimonate. The concentration of RbCl in the entire resulting solution must not exceed 1 M following the addition of the SbCl<sub>3</sub>, since otherwise rubidium chloroantimonate is coprecipitated. The cesium chloroantimonate is then placed in a boat and heated in a stream of HCl gas at 250°C. The SbCl<sub>3</sub> distills off and the CsCl remains as a residue. It can be tested spectroscopically for K and Rb.

The mother liquor from the cesium chloroantimonate precipitate still contains a considerable amount of Cs. It is evaporated to dryness and the residue is freed of  $SbCl_3$  by distillation in a stream of HCl. The resulting mixture of CsCl, RbCl and some KCl is added to the material treated with alcohol to separate the RbCl. To obtain rubidium, the filtrate from the first fractional precipitation of silicomolybdates is used to dissolve the first fraction of the RbCl-KCl precipitate obtained during the separation of CsCl by treatment with alcohol. Next, the silicomolybdate reagent is added until the solution acquires a permanent yellow color, and most of the Rb present completely precipitated. The precipitate is washed four times (vigorous stirring) with 200-ml. portions of 2.5N HCl, filtered through a fritted glass filter, dried in a vacuum desiccator, decomposed in an HCl stream, and is finally freed of SiO<sub>2</sub>, as described above.

The dry residue so obtained consists of RbCl-KCl and is combined with the second and third RbCl-KCl fractions previously obtained in the separation of CsCl by alcohol treatment. The combined residue is redissolved in 180 ml. of 2.5N HCl and once more treated with the silicomolybdate reagent. The latter is added in portions with vigorous stirring until complete precipitation results. The precipitate is washed twice (vigorous stirring) with 60-ml. portions of 2.5N HCl, filtered as above and dried. Finally, it is converted to the chloride and freed of SiO<sub>2</sub>. The purity of the resulting RbCl can be tested spectroscopically.

If the negligible amount of CsCl carried along with the RbCl is not harmful, the latter can be completely precipitated in a single step.

If larger quantities are desired, the wet method of silicomolybdate decomposition is more convenient. The latter complex is treated with barium hydroxide solution saturated at low temperature. (To avoid working with excessive quantities of liquid, part of the Ba(OH)<sub>2</sub> can be added as the solid.) The amount added must exceed by 20% the amount required to decompose the complex into barium molybdate, barium silicate and RbOH (CsOH). The mixture is then boiled for thirty minutes. Flame gases containing  $CO_2$ should not come in contact with the mixture or large amounts of Ba(OH)<sub>2</sub> will be converted to worthless BaCO<sub>3</sub>. The barium molybdate and barium silicate which separate are not filtered off until after the reaction mixture has cooled. The filtrate is then saturated with CO<sub>2</sub> and boiled for fifteen minutes. The BaCO<sub>3</sub> precipitate is filtered off and the filtrate is evaporated with simultaneous addition of hydrochloric acid. The residue contains RbCl and CsCl free of Mo and Ba.

Preparation of silicomolybdate reagent: A boiling solution of 60 g. of NaOH in 400 ml. of  $H_2O$  is prepared, and 172 g. of  $MoO_3$ , free of ammonium salts, is added in small portions over a period of 10-15 minutes. Heating is then stopped and 500 ml. of cold water is poured into the solution. Next, 250 ml. of HNO<sub>3</sub> (d. 1.39) is diluted with water to a volume of 350 ml. and is added. Although the addition should be rapid, only small portions are added at one time. Vigorous agitation must be maintained throughout. No

permanent precipitate should form during these additions. Immediately thereafter, a silicate solution is added in a thin jet and with vigorous stirring. The solution is prepared from 28g. of Na<sub>2</sub>SiO<sub>3</sub>  $\cdot$  9H<sub>2</sub>O dissolved in 125 ml. of 2N NaOH and boiled for 10-15 min.

The deep-yellow silicomolybdate solution is concentrated on a water bath to a volume of 700-800 ml. At this point some ammonium silicomolybdate may separate out if the  $MoO_3$  used was not completely free of the  $NH_4^+$  salt.

Recovery of molybdosilicate solution. The silicate-containing precipitate of barium molybdate is boiled for 30 minutes with a slight excess of sodium carbonate solution, using vigorous stirring. On cooling, the silicate and carbonate are filtered off. A silicomolybdate solution is then prepared from the filtrate, which contains all of the molybdic acid in the form of Na molybdate. The directions are the same as given above. The only difference is that 280 ml. of concentrated nitric acid is required here, instead of 250 ml., because the Na molybdate solution still contains a slight excess of Na<sub>2</sub>CO<sub>3</sub> (caution: violent foaming occurs on addition).

The excess of silicomolybdate reagent, which is added to the HCl solution of carnallite in order to completely precipitate the Rb, can be separated off as yellow ammonium silicomolybdate upon addition of an excess of a concentrated aqueous solution of  $\rm NH_4NO_3$ . Molybdic acid is recovered from the above ammonium salt by the same method as used for processing Rb silicomolyb-date.

II. Other methods. The industrial D'Ans process uses tetraoxalates in the preparation of Rb (and Cs) from carnallites. In this method rubidium carnallite, previously isolated by a series of fractional precipitations, is dissolved in a small amount of water and the hot solution treated with an excess of oxalic acid. This excess should be so large that after cooling and separation of  $RbH_3(C_2O_4)_2 \cdot 2H_2O$  the strongly acid solution (HCl) will still be saturated with oxalic acid. This can be easily checked under a microscope. The well-crystallized tetraoxalate is suction filtered and recrystallized from hot water. The tetraoxalates of Rb and Cs are similar, both being readily soluble in hot water. If it is desired to obtain the Cs, which is present together with the Rb. a brief series of fractional precipitations of the oxalate must be performed. As a result, the Cs, which is the more soluble component, will concentrate in the mother liquor. The Rb tetraoxalate is converted to carbonate by calcination at a moderate temperature, immediately yielding a pure-white product free of traces of Cs.

The precipitation of Rb and Cs in the laboratory is not quite complete. Even though the mother liquor, containing hydrochloric and oxalic acid, can be further processed to obtain the residual amounts of Rb and Cs [together with  $Mg(OH)_2$ ], this can be done effectively only on an industrial scale.

PROPERTIES:

RbCl: Formula weight 120.9. M.p. 717°C, b.p. 1383°C. Solubility (0.55°C) 77.34; (18.70°C) 90.32; (114.0°C) (b.p.) 146.65 g. RbCl/100 g.  $H_2O$ . Solubility in ethyl alcohol (25°C): 0.078 g. RbCl/100 g. alcohol. d. (x-ray) 2.79. B1 structure type.

CsCl: Formula weight 168.4. M.p. 645°C, b.p. 1303°C. Solubility (0.70°C) 162.29; (16.20°C) 182.24; (119.4°C) (b.p.) 289.98 g. CsCl/100 g. H<sub>2</sub>O. d. (x-ray) 3.99. B2 structure type.

**REFERENCES:** 

- G. Jander and H. Faber, Z. anorg. allg. Chem. <u>179</u>, 321 (1929).
   G. Jander and F. Busch, Z. anorg. allg. Chem. <u>187</u>, 165 (1930); <u>194</u>, 38 (1930).
- II. J. D'Ans, Angew. Chem. 62, 118 (1950).

#### CESIUM CHLORIDE AND CESIUM ALUM FROM POLLUCITE

#### CESIUM CHLORIDE

Pollucite (pollux) is a cesium aluminum silicate, about onethird of which is  $Cs_2O$ . Lenher, Kemmerer and Whitford recommend the following method for obtaining Cs from this mineral.

The mineral is thoroughly pulverized and about 5 kg. of it is passed through a fine-mesh flour sieve. Such fine division of the mineral ensures slow but complete decomposition with concentrated hydrochloric acid. After evaporating the hydrochloric acid and dehydrating the silicic acid at  $110^{\circ}$ C, the mass is extracted with 3N HCl (this is the optimal concentration for subsequent precipitation of Cs-Sb chloride) and the Cs is precipitated as cesium antimony chloride,  $3CsCl \cdot 2SbCl_3$ , on adding a slight excess of a solution of SbCl<sub>3</sub> in 3N HCl. A small amount of CsCl remaining in solution can be recovered by evaporating the filtrate, dissolving the residue in 3N HCl, and precipitating with the SbCl<sub>3</sub> solution.

The cesium antimony chloride is hydrolyzed by boiling with water. The solution contains the Cs, a very small amount of Sb, and traces of Fe and Al. The Sb is precipitated with  $H_2S$ ; the CsCl is either obtained directly by evaporating the solution, or else it is converted first to nitrate and then to carbonate (see "Very Pure Alkali Metal Carbonates"). The yield is about 37%, based on the original pollucite.

### CESIUM ALUM

The alkali metals and Al are present in pollucite in about the right proportions for the formation of alum. Hence, the procedure of Clusius and Stern can be followed. The mineral is decomposed with hydrochloric acid and Cs is precipitated as a low-solubility alum by treatment with sulfuric acid. Thus, for example, 0.5 kg. of very finely pulverized (0.01 mm.) pollucite in one liter of 18%hydrochloric acid is evaporated to dryness on a water bath. This procedure is repeated three times, and each time the dry residue is extracted with one liter of  $H_2O + 100$  ml. of concentrated hydrochloric acid. This is followed by suction filtration on a filter cloth. The filtered extracts are combined and concentrated to one liter, and the silicic acid, which separates out almost completely, is decanted. The alum is then gradually precipitated with 200 ml. of concentrated H<sub>2</sub>SO<sub>4</sub>. After cooling, about 545 g. of crude yellowish alum is obtained. The mother liquor is practically free of Cs. The crystallization is repeated several times, in each case dissolving 250 g. of the alum in 2.5 liters of boiling water in a four-liter Erlenmeyer flask. On slow cooling (constant agitation, 10 hours) the alum again separates out. The material obtained after six crystallizations shows no traces of other alkali metals. One way to ensure pure alum is to check the purity of the mother liquor from which it is precipitated. The specific conductivity of a pure Cs alum solution is  $1.39 \cdot 10^{-3} \Omega^{-1} \cdot \text{cm.}^{-1}$  (measured at 25°C, saturated solution).

**REFERENCES:** 

- V. Lenher, G. Kemmerer and E. Whitford, Ind. Eng. Chem. <u>16</u>, 1280 (1924).
- K. Clusius and H. Stern, Helv. chim. Acta 33, 462 (1950).

## Free Alkali Metals

ELECTROLYTIC PREPARATION OF LITHIUM

$$\begin{array}{rcl} \text{LiBr} = \text{Li} + \frac{1}{2} \text{Br}_2 \\ 86.9 & 6.9 & 79.9 \end{array}$$

Pure Li is prepared (via the method of Ruff and Johannsen) from LiBr which is melted in an electric arc in the presence of 10-15% LiCl (the LiBr is obtained from Li<sub>2</sub>CO<sub>3</sub> by evaporating the latter from hydrobromic acid). Fig. 260 is a scale drawing (1:5) of the Muthmann electrolysis vessel used for the melting procedure. It is made of copper and its upper part is cooled with water. While the melting point of pure LiBr is about  $546^{\circ}$ C and that of LiCl is  $606^{\circ}$ C, a mixture of LiBr with 13% LiCl solidifies at  $520^{\circ}$ C. The electrolysis proceeds at 10 v. (as measured across the terminals)



Fig. 260. Electrolytic preparation of lithium. The electrolysis vessel, made of copper, has a diameter of about 8 cm. at the top and about 2.5 cm. at the bottom.

and 100 amp. A graphite rod is used as an anode and two 4-mm. iron wires serve as cathodes. The metal, which deposits at the cathodes, is scooped up from time to time with a flat iron spoon and, while still liquid, is separated from the solidified melt on a cold stone plate. It is next freed of adhering salt using Borchers' method, i.e, by immersion in a paraffin bath (180-200°C). The salt settles to the bottom. while the metal rises to the surface. After cooling, it is washed with ligroin. It is stored under ligroin (d. 0.56) in completely filled, tightly closed vessels.

**REFERENCES:** 

 O. Ruff and O. Johannsen, Z. Elektrochem. <u>12</u>, 186 (1906).
 W. Borchers, ibid. 3, 39 (1895).

### PURIFICATION OF TECHNICAL GRADE LITHIUM: REMOVAL OF POTASSIUM

Technical grade lithium prepared by electrolysis of a fused, low-melting mixture of LiCl and KCl is, according to Ruff and Johannsen, contaminated with a few percent of potassium. The latter is removed by the Guntz and Broniewski procedure. The Li is converted to LiH by heating in a stream of  $H_2$  at 700-800°C (see the preparation of LiH). The potassium volatilizes in the metallic form, since its hydride is unstable at this high temperature. The LiH is then decomposed under vacuum at 1000°C, the purified Li being condensed on a water-cooled iron cylinder mounted in the reaction vessel.

**REFERENCES:** 

O. Ruff and O. Johannsen, Z. Elektrochem. <u>12</u>, 186 (1906). A. Guntz and W. Broniewski, J. Chim. phys. <u>7</u>, 468 (1909).

ALKALI METALS OBTAINED BY REDUCTION WITH ZIRCONIUM

Dissolved gases trapped during preparation are particularly difficult to remove from alkali metals. The removal of these gases

requires repeated distillation under vacuum, and even then the complete separation is difficult to achieve. Therefore, laboratory methods for preparation of pure alkali metals should avoid any contact of gases with the nascent metal. According to De Boer, and also Broos and Emmens, the reduction of alkali chromates as well as of bichromates, molybdates and tungstates with zirconium powder meets the above specification, particularly well.

To prepare pure Cs, Rb or K, one part by weight of  $Cs_2CrO_4$  (Rb<sub>2</sub>CrO<sub>4</sub> or K<sub>2</sub>CrO<sub>4</sub>) is mixed with four parts by weight of fine Zr powder. The mixture is compressed into rods and heated in a highly evacuated, thoroughly preheated quartz tube (or other suitable apparatus).

The reactions start smoothly at  $725^{\circ}$ C (700 or  $800^{\circ}$ C). The heating is continued until a temperature of  $1000^{\circ}$ C is reached. The alkali metals form oxide-free shiny, mirrorlike deposits on the colder parts of the tube. Yields: Cs 90-96%; Rb practically quantitative; K up to 80%.

To prepare Li it is necessary to mix  $\text{Li}_2\text{CrO}_4$  with eight parts by weight of Zr to prevent explosive reduction between 450 and  $600^{\circ}\text{C}$ . The yield of the metal is very low.

In cases where, for reasons associated with the limitations of equipment, the temperatures required for the preparation of pure Cs, Rb and K from chromates cannot beused, bichromate mixtures consisting of one part of  $Cs_2Cr_2O_7$  (Rb<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub>, K<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub>) to ten parts of Zr must be used. A smooth reduction will start at about 380°C (370°, 380°C). The yield of Rb is 80-90%. It is free of oxide, as is the potassium formed in this reaction. The Cs product, how-ever, contains some oxide, but, if the mixture ratio is changed to one part of  $Cs_2Cr_2O_7$  to 20 parts of Zr, the Cs will be free of oxide.

To prepare pure Na it is best to replace the deliquescent chromate or bichromate by  $Na_2MOO_4$  or  $Na_2WO_4$ , which is mixed with four parts by weight of Zr powder. Evolution of sodium vapor begins smoothly at about 550° or 450°C, respectively.With  $Na_2MOO_4$ the yield is practically quantitative, and with  $Na_2WO_4$  it is 80%. The Na is free of oxide. For the preparation of Rb and Cs chromates or bichromates, see the section on chromium. According to the following equations,  $\text{Li}_2\text{CrO}_4$  is formed on boiling a solution of  $(\text{NH}_4)_2\text{Cr}_2\text{O}_7$  with LiOH;  $\text{Na}_2\text{MoO}_4$  and  $\text{Na}_2\text{WO}_4$  are obtained by the reaction  $\text{Na}_2\text{CO}_3$  with  $\text{MoO}_3$ , or  $\text{WO}_3$ respectively.

 $\begin{array}{rl} 4 \operatorname{LiOH} + (\mathrm{NH}_4)_2 \mathrm{Cr}_2 \mathrm{O}_7 &= 2 \operatorname{Li}_2 \mathrm{CrO}_4 + 2 \operatorname{NH}_3 + 3 \operatorname{H}_2 \mathrm{O} \\ & 95.8 & 252.1 & 259.8 & 34.1 & 54.0 \\ \mathrm{Na}_2 \mathrm{CO}_3 &+ \operatorname{MoO}_3(\mathrm{WO}_3) &= \operatorname{Na}_2 \mathrm{MoO}_4(\mathrm{Na}_2 \mathrm{WO}_4) + \operatorname{CO}_2 \\ & 106.0 & 143.9 \ (231.9) & 205.9 & (293.9) & 44.0 \end{array}$ 

**REFERENCES:** 

J. H. De Boer, J. Broos and H. Emmens, Z. anorg. allg. Chem. 191, 113 (1930).

#### ALKALI METALS OBTAINED BY DECOMPOSITION OF AZIDES

Suhrmann and Clusius succeeded in preparing very pure, gas-free alkali metals by thorough decomposition of their azides under high vacuum.

The equipment used (Fig. 261) must be Pyrex which is not seriously attacked by Na, K, Rb, Cs or Li.

When "physically pure" alkali metals are to be prepared, all the glass parts of the equipment are joined by fusion and there



Fig. 261. Preparation of alkali metals from azides. 7) Decomposition tube;  $v_1$ ,  $v_2$ ) receivers; a) ampoule for storing the metal; b, c)fused joints;  $k_1$ ,  $k_2$ ) cooling traps; q) mercury valve.

are no greased stopcocks in that part of the apparatus which is under vacuum. In other cases, where purity requirements are less vigorous, the use of ground jcints and stopcocks is permissible. These, however, must not come in contact with the liquid alkali metal or its vapor.

The azides are finely pulverized in an agate mortar and placed in a retort r (amounts: 10-12 g. of NaN<sub>3</sub> or KN<sub>3</sub>, 6-7 g. of RbN<sub>3</sub> or CsN<sub>3</sub>). The salt, which fills about one quarter of the retort, is then distributed over the entire tube. However, RbN<sub>3</sub> and CsN<sub>3</sub> are not placed directly in the retort. Instead, a quartz tube sealed at one end is filled with the salt and inserted into the retort. Next, ris sealed onto the rest of the apparatus and the latter is evacuated by means of a high-speed forepump. (In the original preparation, a mercury diffusion pump made of glass was used.) At the same time the apparatus is thoroughly heated to remove traces of gases. The cooling tubes  $k_1$  and  $k_2$  are kept immersed in liquid nitrogen until the end of the experiment. An electric heater is pushed over r and a temperature of 200°C is maintained for 12 hours, while the apparatus is continuously evacuated. A Geissler tube, connected to the apparatus as a vacuum gauge, must always show a high vacuum (ready discharge). Next, the furnace temperature is gradually raised and the mercury value q is closed, to avoid entrainment of the azide by the free N<sub>2</sub> during the subsequent decomposition and its deposition in receiver  $v_{2}$ . The decomposition temperatures of the azides are: NaN<sub>3</sub>, 275°C; KN<sub>3</sub>, 355°C; RbN<sub>3</sub>, 395°C (quartz tube); CsN<sub>3</sub>, 390°C (quartz tube). Decomposition of NaN<sub>3</sub> begins before the melting point is reached; KN<sub>3</sub> melts at 343°C, RbN<sub>3</sub> at 321°C, CsN<sub>3</sub> at 326°C. At the start of the decomposition, the temperature is adjusted in such a way that pressure in the apparatus is not above 0.1 mm. To prevent a sudden pressure rise, a surge vessel is connected to the apparatus via a stopcock. This is an eight-liter vessel, thoroughly evacuated by means of the forepump. To assure rapid pressure relief all tubing must be 12 to 16 mm. in diameter. The decomposition sometimes does not start until 3-4 hours after the proper temperature is reached (especially in the case of KN<sub>3</sub>). The azides, should not be overheated because an explosive decomposition may occur and the apparatus destroyed.

The end of the decomposition is indicated by cessation of discharge of the Geissler tube (high vacuum). As the N<sub>2</sub> pressure decreases, the alkali metal formed is distilled from r into collector  $v_1$ . The retort r is then sealed off at point b and the stopcock to the surge vessel is closed, while the forepump is still operating. The high-vacuum pump (in the original, a mercury diffusion pump made of glass) is then started and the mercury valve q is opened. Then the connection to the forepump is sealed off at c. The pressure in the apparatus is reduced to below 10<sup>-7</sup> mm. Under these conditions the alkali metal readily distills from  $v_1$  to  $v_2$  on slight heating with an electric heater. The metal in  $v_2$  is then melted so that it flows into ampoule a, which is eventually sealed.

In more successful work the decomposition is completed in 3 to 4 days as described provided the pressure is not allowed to rise above 0.1 mm.; in less successful preparations it takes 6 to 8 days. The alkali metals thus obtained are completely free of gas, so that no pressure rise is observed on heating them under high vacuum.

The yield of these alkali metals is approximately 100% for NaN<sub>3</sub>, 80% for KN<sub>3</sub>, 60% for RbN<sub>3</sub>, and 90% for CsN<sub>3</sub>. The residue is light brown in the case of KN<sub>3</sub>, blue-green in the case of RbN<sub>3</sub>, and yellowish-gray in the case of CsN<sub>3</sub>. It consists of the nitride mixed with silicate and undecomposed azide.

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R. Suhrmann and K. Clusius, Z. anorg. allg. Chem. 152, 52 (1926).

RUBIDIUM AND CESIUM OBTAINED FROM THE CHLORIDES

RbCl (CsCl) +  $\frac{1}{2}$ Ca =  $\frac{1}{2}$ CaCl<sub>2</sub> + Rb (Cs) 120.9 (168.4) 20.0 55.5 85.5 (132.9)

This method, proposed by Hackpill, uses the apparatus shown in Fig. 262. A weighed amount of RbCl (CsCl) is thoroughly predried in an oven at about  $150^{\circ}$ C and mixed with the required amount of Ca turnings. An iron insert crucible *b* is filled with the mixture and stoppered with a plug *c* made offine steel wool. A ratio of 10 g. of RbCl (15 g. of CsCl) to 8 g. of Ca has proven satisfactory. The apparatus (Fig. 262) with a reaction vessel 30-35 mm. in diameter, will hold a maximum of about 35 g. of RbCl (50 g. of CsCl), together with the appropriate amount of Ca.



Fig. 262. Apparatus for preparation of metallic rubidium (or cesium) by reduction of the chloride with calcium.

After reaction vessel a has been opened, the filled crucible is placed in it and the vessel sealed directly above the side tube (dotted line in figure), leaving as little dead space as possible. The reaction vessel is placed inside the heater o and its top is wrapped as well as possible in asbestos wool, to prevent cold spots where later on the alkali metal vapor could condense. Heating under high vacuum is then started. The temperature is checked with a thermocouple, protected by a thin ceramic sheath and inserted between the vessel wall and the heater. The thermocouple sheath must not touch the vessel wall and is prevented from adhering to the latter by a few tufts of asbestos.

As the temperature rises to  $250^{\circ}$ C, large quantities of gases are released from the mixture. It is sometime before the vacuum pump is able to remove these. During this time all glass parts of the apparatus, are uniformly heated by fanning with a gas flame. After 2-3 hours, when the evolution of gas subsides, the temperature is raised further until all of the alkali metal slowly distills into receiver e. The distillation lasts 2-3 hours. Toward the end the temperature in the reaction vessel rises to about  $650^{\circ}$ C. The reactor is then disconnected from the remainder of the apparatus by melt-sealing constriction d while vacuum is maintained. This must be done before the heating is shut off, since vessel a is deformed by heat and will usually break on cooling.

The crude metal is distilled from collector e, using a tubular heater. The liquid metal runs through the ground joint and through constriction  $\mathcal{J}$  into the second receiver g. (The ground joint should be carefully greased to prevent grease spillover into the passage.) The temperature in this vessel is considerably lower and the vacuum higher than in the first distillation stage. If very pure, the twice distilled metal will not wet the glass walls. Receiver qis finally sealed off under vacuum at constriction f. It thus becomes free and can be tipped in such a way that the metal, melted with low heat, will flow into ampoules i attached on the side. The number and size of these ampoules can be varied as desired. The liquid metal filling the ampoules will solidify more rapidly if externally cooled with a piece of Dry Ice. The ampoules are finally filled with very pure  $N_2$  through stopcock h and sealed off at the constriction points in their connecting tubes. If the ampoules are weighed before and when filled, the weight of the contents can be determined to 0.01 g.

Somewhat larger quantities of alkali metal can be prepared in a stainless steel reactor (Fig. 263). A steel pipe l (inside diameter about 38 mm.), is filled with the reaction mixture m, covered with a steel wool plug n and closed with a heavy steel cap. The latter is sealed on with a lead gasket covered with a very thin copper foil sheath to protect it against corrosion by the alkali metal vapor. Just as in the case of the glass apparatus, the upper part of the reactor (including the cap) must be thoroughly insulated. The metal vapor condenses in the water-cooled section of the



Fig. 263. Steel reactor for preparation of metallic rubidium (or cesium) by reduction of the chloride with calcium. side arm pipe. The cooling should not be too intensive and the liquid metal should be able to flow (without solidifying) into the next part of the apparatus via ground joint p. This remaining part of the apparatus is made of glass and, beginning with constriction d, is identical to that shown in Fig. 262.

Such steel reactors allow processing 100 g. of RbCl, or 150 g. of CsCl in a single run. The yield of double-distilled metal is 90-98%.

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- G. Brauer, private communication.

PROPERTIES OF ALKALI METALS

Atomic weight		M.p.°C	B.p.℃	d (18°C) Hardness (mohs	
Li	6.940	179.0	1336	0.534	0.6
Na	22.997	97.8	883	0.97	0.4
K	39.096	63,5	762	0.86	0.5
Rb	85.48	39.0	696	1,52	0.3
Cs	132.91	28.6	670	1.89	0.2

Very soft. Fresh surface is silver-white. Highly reactive. Immediately form hydroxide-carbonate crusts in air, usually accompanied in the case of Cs, by ignition. Water and alcohol are decomposed with liberation of  $H_2$ . (Storage, see below.) A2 structure type.

### PURIFICATION OF ALKALI METALS BY VACUUM DISTILLATION

(See also the introduction to the section on Preparation of Alkali Metals by Reduction with Zirconium.)

Fig. 264 shows Brauer's apparatus for distilling Cs and Rb. Ordinary chemical glassware may be used. (However, according to Hevesy and Lögstrup, distillation of potassium requires Pyrex apparatus.) Tube B is slightly inclined. The ampoules are

melt-sealed to D in such a way that they are horizontal during the early distillation stage. The apparatus is initially sealed off at



Fig. 264. Distillation of cesium (also rubidium, potassium, sodium). Scale 1:7. In the actual arrangement the ampoules, sealed to tube D, lie in a plane perpendicular to that of the drawing; hence, they remain horizontal during the distillation and do not become prematurely filled with liquid cesium. Tube B is likewise not quite coaxial with D but is inclined slightly downward.

point A and evacuated by means of a mercury diffusion pump connected to the ground glass joint F. the evacuation continues for several hours, during which all of the apparatus, from A to E, is dried by heating to 400-500°C. The system is then filled (through F) with pure dry  $N_2$ . The tube end is broken off at A and a boat containing benzine-covered cesium is introduced into tube B. The metal should be free from any incrustations and washed in light benzine (mineral spirits). The cesium in the boat must at all times be protected either by the benzine layer or by a blanket of dry  $N_{2}$ , or by both, as in the present example, where the benzine-covered metal is pushed into the tube from which a stream of N<sub>2</sub> is issuing. Following the insertion of the boat, the benzine is volatilized, B is resealed at A, and the apparatus is evacuated to at least  $10^{-4}$  mm. Next, an electric oven is placed around B, reaching close to C, and the metal is heated until all of it distills into D. Simultaneously, the U tube E is cooled with liquid nitrogen or with Dry Ice-acetone mixture to keep Hg and oil vapor away from the metal. Where purity is not critical, the U tube can be dispensed with. Following the distillation, the molten metal in tube D can be distributed among the storage ampoules by rotation of the apparatus. After cooling, the apparatus is refilled with N<sub>2</sub> and the ampoules are sealed off.

If the metal must be distilled several times, several B tubes are sealed on to D.

The vacuum distillation of Li is carried out according to method of Remy-Genneté, using the same procedure as for Ca, Sr or Ba. The apparatus in this case is a vertical tube. An iron crucible with the metal is inserted and the latter is allowed to distill from the crucible onto a cold finger hanging above. An apparatus of this type, improved by Ehrlich, is shown in Fig. 259 and is discussed under the distillation of Ca.

To prepurify the Li by removing the more volatile impurities, the first fraction is taken off by heating a fairly long time at a relatively low temperature. The apparatus is then cooled under vacuum and opened, and the first fraction of the metal is discarded. The actual distillation is then performed at a higher temperature. The charge should not be distilled to dryness. A residue of Li, containing high-boiling impurities, should be left behind in the crucible.

Lely and Hamburger describe a similar apparatus for the distillation of Na.

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### STORAGE AND HANDLING OF ALKALI METALS BEFORE USE

Lithium is stored in tightly closed vessels, completely filled with petroleum ether. Sodium is usually stored under kerosene. To clean the surface before use, the metal pieces are dried with filter paper, treated with absolute alcohol, and washed with pure petroleum ether. Potassium is usually also stored under kerosene. According to Wislicenus, Elvert and Kurtz, rolling potassium shot around under ether containing a few drops of alcohol will remove the brown crust from the surface. Rubidium and cesium are generally stored under paraffin oil, since they react rapidly under kerosene. Before use these metals are washed with petroleum ether or benzene, thoroughly predried with Na, to remove the oil. The solvents are evaporated in a current of dry  $CO_2$  or removed under vacuum.

To remove oxide-hydroxide-carbonate crusts from Na and K, the following simple treatment has been suggested by Bornemann: a clean wire screen with 1-mm. openings, is inserted into a melting tube 50 cm. long and 15-20 mm. in diameter. The tube is sealed at one end and constricted in the middle to a diameter of 3-4 mm. The constricted section should be very short. The screen is bent to a hemispherical shape so that it fits snugly in the tube. It is pushed down the tube until it reaches the constriction. The tube

above the screen is half filled with freshly cleaned, well-dried pieces of Na or K. The tube is then sealed at a point about 20 cm. above the constriction. Over several hours the metal will have absorbed all the H<sub>2</sub>O, O<sub>2</sub> and CO<sub>2</sub> from the air in the tube and will thus be essentially under an N<sub>2</sub> blanket. To check this, the metal is melted and left to resolidify, while the tube is horizontal (so that the metal will not run through the sieve). This procedure is repeated until the bright metal surface ceases to dull, even after a considerable time, indicating that the last traces of O<sub>2</sub> have been absorbed. The tube is then set vertically and its lower end heated to a temperature above the melting point of the metal so that the latter slowly melts. It gradually runs through the sieve, while the oxide, hydroxide and carbonate are left behind. Should the constriction become plugged, this spot is not heated. Instead, the flow is helped along by lightly tapping the tube against a soft support. At the end of filtration the metal is sealed in lower part of the tube by melting the constriction. No metal vapors must be allowed to form during sealing since they might attack the glass as well as cause pressure inside the tube.

A hopper designed by Zintl, Goubeau and Dullenkopf (Fig. 265) is used to fill small thin-walled glass spheres with high-purity Na

(or K. etc.). The sphere is placed in the hopper and predried under vacuum for a long time. It is next weighed, together with its long, narrow capillary. Finally, it is put back into the hopper, with the capillary opening directed downward, as shown in the figure. The hopper is evacuated via ground joint adapter a. This takes a long time and high vacuum must be used. Simultaneously with the evacuation, the lower half of the hopper is heated to about 200°C on an oil bath to remove the water film lining the inner wall surface of the sphere. After cooling, pure dry N<sub>2</sub> is allowed to flow through a, the ground cap b is lifted momentarily, and a compact chunk of distilled Na (or K, etc.) is quickly inserted and placed as indicated in the figure. The metal should be as free from oxide as possible. The evacuation is immediately repeated. After a while, the heating on the oil bath is resumed in order to slowly melt the metal. Bright metal flows into the attached



Fig. 265. Filling of glass spheres with oxide-free alkali metal.

bulb c leaving behind all of the oxide in the form of a continuous film. Enough pure  $N_2$  is then carefully let in to force the liquid metal into the sphere k so that its lower half is filled. Next, sufficient  $N_2$  is pumped out to siphon the metal back into the capillary

down to level s, thus forming a discontinuity. The metal will be retained in the sphere only if the latter has been thoroughly predried or it will empty as the N is pumped out. The metal is cooled until solid and the N pressure in the hopper is raised to 1 atm. The cap b is then removed and bulb c is immersed in a warm bath. As soon as the metal in c has melted, the sphere is withdrawn with glass hooks. This is done rapidly, while the capillary is still plugged with solid material. The capillary is then melt-sealed at the metal-free discontinuity close to the sphere. The filled sphere is weighed together with the cleaned capillary.

Sodium residues, which are still usable, are melted under toluene or xylene. Small amounts of residual sodium are rendered unreactive by being placed in portions in alcohol containing only a small amount of water.

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- G. Bornemann, Angew. Chem. 35, 227 (1922).
- E. Zintl, J. Goubeau and W. Dullenkopf, Z. phys. Chem. A <u>154</u>, 21 (1931).

### SODIUM DISPERSIONS IN INERT LIQUIDS

Dispersions of molten Na (d. at  $100^{\circ}$ C, 0.928) in inert liquids containing about 50% metallic Na (particle size 1-20 millimicrons) can be prepared in the laboratory by mechanical dispersion. The boiling point of the inert liquid must be higher than the melting point of Na (97.5°C). Such liquids include toluene, xylene, some light mineral spirits, kerosene, heptane, n-octane, mineral oil, and naphthalene. Appropriate agents (0.25-1%) contribute to the reduction of particle size (soot, copper powder, pyridine, etc.) or stabilize the dispersion (oleic acid, aluminum stearate, calcium stearate, etc.).

The usual safety devices generally employed in working with Na and with flammable solvents (such as goggles, safety shields, gloves, purged and sealed heaters, etc.) must be used in the preparation of  $N_2$  dispersions. In case of fire, the burning Na should be covered with dry soda. Carbon tetrachloride extinguishers should not be used. Carbon dioxide extinguishers may be used in fighting solvent fires.

Disk-shaped stirrers with beveled teeth\* (see Fig. 266c) as well as turbine-type agitators\*\* which entrain the liquid along the

<sup>\*</sup>Manufactured by Cowles Dissolver Co., Cayuga, N.Y., and others.

<sup>\*\*</sup>Manufactured by Premier Mill Corp., Geneva, N.Y., and others.

axis and eject it centrifugally through narrow slots (Fig. 266*p*) are effective devices for dispersing Na.

The U.S. Industrial Chemicals Co. has suggested using a twoliter vessel, (diameter about 15 cm.) with a disk stirrer about 7.5 cm. in diameter, rotating at 4000-6000 r.p.m. (see Fig. 266). With an agitator diameter of 2.5 cm. and a speed of 8000-15,000 r.p.m., a half- to three-liter flask is recommended.



Fig. 266. Vessel and stirrers for Na dispersion in inert liquids.

A cleaned and dried apparatus is charged with 400 g. of dry dispersing medium and the substance selected as a dispersion aid is added to the latter with slow stirring. The air is displaced with nitrogen, the stirring is stopped, and 400 g. of bright Na shavings is introduced into the apparatus. The apparatus is then again flushed with  $N_2$ . All of the Na melts on heating to  $105^{\circ}$ C, and the stirrer is restarted and brought slowly to top speed.

As the particle size decreases, the mixture turns a deeper shade of gray. In most cases, the particle size drops to 10-15 microns within 10-15 min. If the particles still appear too large when examined under a microscope, the stirring is continued for another five minutes. The dispersion is then left to cool to or below  $80^{\circ}$ C without any stirring. The stirring can be resumed below  $80^{\circ}$ C since the particles will not recoalesce. When the dispersion is at room temperature it may be poured into dry storage vessels. Sodium dispersions should be stored in tightly closed vessels under a nitrogen blanket. Air or moisture destroy the dispersion and may easily cause a fire. For safety the glass storage vessels should be placed inside a protective vessel and embedded in diatomaceous earth. Iron vessels are also worth considering as storage containers. Storage vessels must be free of Na traces both on the outside and in the area of the stopper. Filter paper used to remove the sodium remnants must be well impregnated with kerosene. Dry paper, linen and the like ignite as soon as they come in contact with Na dispersions.

Where less concentrated dispersions are used, it is best to dilute the stock just before use. The dilution fluid should have a boiling point below the melting point of Na.

After emptying the apparatus the reactor is first rinsed with kerosene and then treated with water vapor (which must be free of liquid water) to react the last traces of Na.

REFERENCES:

Sodium Dispersions, U.S. Industrial Chemical Co., 1957. V. L. Hansley, U.S. Patents 2,394,608 (1946) and 2,487,334 (1949).

### FINELY SUBDIVIDED SODIUM ADSORBED ON INERT SOLIDS

Molten Na spreads spontaneously on the surface of inert solids at 100 to 200°C in a N<sub>2</sub> atmosphere; monatomic sodium layers can thus be achieved. Suitable solid substances are NaCl, Na<sub>2</sub>CO<sub>3</sub>, carbon (charcoal), metal powders,  $Al_2O_3$  and SiC. In some cases the reaction products prepared with the aid of finely subdivided metallic Na themselves prove to be effective carriers. Carrier materials consisting of fine particles coated with metallic Na remain free flowing over a wide range of temperatures and concentrations. Depending on its grain size, common salt will adsorb 2-10% Na. Soda adsorbs 10% Na, aluminum oxide 20-25%, activated carbon 30%. These substances remain free flowing up to the melting point of Na. At high Na contents, the materials convert to pastes.

The table below gives the optimum dispersing temperatures, the contents of finely subdivided Na and the appearance of the mixtures with some carrier materials.

According to a laboratory manual published by the National Distillers Chemical Co. a well-dried three-necked Pyrex flask, capacity 1-3 liters, may be used as the reactor (see Fig. 267). A stirrer with a graphite-packed gland is inserted through the middle neck. The other two necks serve for filling and temperature measurement.

Carrier material	Optimal dispersing temperature, C	Sodium content, %	Appearance
Activated alumina Activated coconut charcoal Iron powder ( $150\mu$ ) Colloidal carbon Common salt ( $180-420\mu$ ) White sand ( $420-840\mu$ ) Calc. soda ( $50-150\mu$ ) Zirconium oxide ( $50\mu$ )	140-160 120 150-200 170 150 150-165 150 250	20-25 ≫35 >30 2-10 5 10 10	black silver to black (pyrophoric) gray black (pyrophoric) gray to black gray gray to black dark gray

A tube on each neck serves as the inlet and outlet for the nitrogen. The flask is heated by means of a tightly fitting electrical heating mantle. A dish is placed below the apparatus as a safety pan in case of breakage. An oil bath may also be used for heating. The same safety measures as those mentioned in the preceding section must be observed.

The flask is charged with, for example, 300 g. of dried, calcined sodium carbonate and the air is displaced with dry N<sub>2</sub>. By stirring at 100-300 r.p.m., the soda is whipped to about twice the original volume, while it is heated to  $150^{\circ}$ C. Then 10 g. of Na (in pieces weighing 2-5 g.) is added through the filling neck. As soon

as the Na melts the stirring is accelerated and the high speed is maintained for about five minutes. As the Na distributes over soda, the latter changes from white to gray.

Wherever possible, finely subdivided Na is used directly following its preparation, and the reaction for which it is intended is carried out in the same vessel in which the Na dispersion has been prepared. If, however, the dispersion must be stored in another container for future use, the transfer should be done under  $N_2$ , after precooling of the apparatus in a nitrogen stream. Well-dried metal storage containers should be used for this purpose. They must be kept in a dry place and away from flammable materials.

To decompose Na finely dispersed on solid materials, it is burned in some safe spot, inside an open iron vessel. If neces-





sary, a little kerosene may be added beforehand to produce complete combustion. Small residues of fine Na remaining in the reaction vessel are reacted with dry steam (or rapid stream—use a protective shield) in the reactor previously flushed and filled with N<sub>2</sub>. REFERENCES:

High Surface Sodium, National Distillers Chemical Co., Ashtabula (Ohio) 1953.

### Alkali Hydrides

#### NaH, KH, RbH, CsH and LiH

Li (Na, K, Rb, Cs)  $+ \frac{1}{2}H_2 =$  LiH (NaH, KH, RbH, CsH) 6.9 (23.0; 39.1; 85.5; 132.9) 7.95 (24.0; 40.1; 86.5; 133.9)

NaH, KH, RbH, CsH

The hydrogenation apparatus shown in Fig. 268 is designed to prepare NaH, KH, RbH and CsH following the procedure of Zintl and Harder. A seamless steel liner tube a is inserted into a quartz or Vycor tube r as protection against corrosion by alkali metal vapors. Liner a is preheated for many hours in moist  $H_2$  at 900°C to decarbonize the steel. As iron boat c is placed in sheet iron cylinder b, which is closed on one side. After prolonged evacuation of the apparatus the surfaces of a, b and c are deoxidized by passage of electrolytic  $H_2$  (inlet at h and outlet at the loosely fitting joint  $s_1$ ) and simultaneous prolonged heating of a, b and c to dull red heat, using an electric oven.

The electrolytic  $H_2$  must be very thoroughly freed of  $O_2$  and  $H_2O$  and before entering the reaction zone must pass through a large U tube filled with resublimed  $P_2O_5$ .

Sodium (or potassium) must be freed from adhering highboiling hydrocarbons prior to use. To this end, they are remelted repeatedly under xylene and, when as oxide-free as possible, are placed in boat c.

Rubidium and cesium can be prepared in the reactor itself (see section on Alkali Metals Obtained by Reduction with Zirconium, p.957). In this procedure, the boat is charged with a mixture of  $Rb_2CO_3$  or  $Cs_2CO_3$  and magnesium powder, using a mole ratio of 1:3 (weight ratios are 231.0:73.0 or 325.8:73.0, respectively). The mixture is predried under vacuum at 150°C. The apparatus is evacuated and cylinder b with boat c are slowly heated. In the case of Na and K, the temperature is raised to  $300-350^{\circ}C$  (at which temperature the metals distill). In the case of Rb and Cs, the temperature is  $620^{\circ}C$ . The alkali metals condense inside the steel liner at a. After cooling, tube b with boat c containing impurities and/or residues are pulled out from the reaction zone (still under a  $H_2$  stream) through ground glass joints  $s_2$ . With manometric valve v reconnected, the air is displaced with  $H_2$  and the alkali metal is slowly evaporated by heating at  $300-400^{\circ}C$ and 1 atm. in a stationary hydrogen atmosphere. The hydride formed under these conditions is deposited on both sides outside the heated zone, mostly in the form of cottonlike clusters of colorless, crystalline needles. If the evaporation of the metal is too rapid, the hydride becomes contaminated with condensed metal. From time to time the tube is refilled with  $H_2$  to keep the pressure at 1 atm. When the manometric valve v shows no further pressure drop over a period of 24 hours, the hydride is removed (under a hydrogen stream) from a by pushing it into tube d with a small Pt scoop sealed onto a long glass rod and introduced through  $s_1$ . Tube r is then removed and  $s_3$  is closed off. The hydride can then be transferred to other containers by opening  $s_4$ , under a stream of  $H_2$ .



Fig. 268. Preparation of alkali metal hydrides; r) quartz or Vycor reaction tube;  $\alpha$ ) protective iron liner; b) boat-shielding iron cylinder; c) iron boat.

A finely subdivided NaH suspension may be prepared following a procedure suggested by Ziegler, Gellert, Martin, Nagel and Schneider.

Metallic Na and a dispersing medium are heated at  $200-220^{\circ}$ C with brisk stirring, using an autoclave provided with a magnetic stirrer (rotary and rocking autoclaves are less suitable). At the same time, electrolytic H<sub>2</sub> is forced in from a steel cylinder or with a compressor. Hydrogenation takes place at all pressures. The higher pressure level is important only insofar as it determines the rate of H<sub>2</sub> uptake.

For each liter of reaction volume, 500 ml. of dispersing agent and 75 g. of Na are used. Suitable dispersing media include hexane, heptane, octane (alone or in mixtures), cyclohexane, methylcyclohexane and ethylcyclohexane. It is best to use a dispersing medium with a critical temperature above 200°C. Aromatic media cannot be used since NaH is a very active hydrogenation catalyst at high temperatures, and thus  $H_2$  would be lost through hydrogenation of the dispersing medium.

When the  $H_2$  uptake ceases, the coarse-grained suspension of NaH may be removed from the autoclave. If the suspension is then ground in a ball mill, its color changes sharply from white to gray-black. The reason for this is that the residual metallic Na in the product becomes finely subdivided. In such cases, the hydrogenation must be repeated, as above. The final pure white suspension will retain its color even after wet grinding. If, following the first hydrogenation, the heating is continued for 2-3 hours at 280-300°C under compressed  $H_2$ , the suspension will remain white even upon first wet grinding. Repeated hydrogenation for such material is superfluous.

LiH

Zintl and Harder prepared LiH in a boat made of electrolytic iron and charged with shiny Li under Ar. Since molten Li diffuses through iron, a second electrolytic iron liner is inserted into steel tube a. The hydrogenation proceeds rapidly at 600°C and is complete at 700°C. At this temperature, the LiH product is liquid (m.p. 680°C). On cooling, it becomes coarsely crystalline and appears completely colorless and transparent. The boat is pushed into tube d with a long rod (see above) and ground joint  $s_1$  is shut. A rotary steel milling cutter, about 5 mm. in diameter (see section on Intermetallic Compounds), is then inserted at  $s_5$  in order to pulverize the hydride. A bulge in the lower half of tube dprovides the necessary support for the boat during this operation.

Following the Albert and Mahé procedure, LiH is prepared in quantities of 1 kg. in a low-carbon steel pot, externally protected from scaling with an aluminized steel jacket. The upper part of the pot and its flat, rubber-gasketed lid are water cooled. The lid has nozzles for  $H_2$  input, a vacuum connection, and a thermocouple. Two concentric cylindrical "Armco" iron crucibles are placed inside the pot. These fit snugly inside each other and in the pot.

For 1 kg. of LiH (about 890 g. of Li) the innermost crucible should measure 125 mm. in diameter, 350 mm. in height and have walls 2 mm. thick.

The Li is introduced, and the apparatus is evacuated, filled with  $H_2$  and heated. Hydrogen uptake starts at 500°C and becomes vigorous at 650°C. A steady pressure of 0.25 atm. gauge is main-tained; heating above 700°C must be avoided. The reaction time is about three hours. After the complete cooling, the LiH is taken out under a blanket of  $CO_2$  to prevent spontaneous ignition of readily oxidizable sublimates which deposit on the cold parts of the apparatus.

The entire operation lasts about eight hours and yields wellcrystallized, hard LiH about 99.6% pure. It is bluish in spots due to contamination with a slight excess of Li.

PROPERTIES:,

Colorless substances, decomposed by moisture. Stability to  $O_2$  decreases sharply from LiH to CsH: LiH reacts only at red heat; NaH ignites in  $O_2$  at about 230°C; KH, RbH and CsH react at room temperature. Equilibrium hydrogen pressure for LiH is 0.023 mm. at 23.5° and 70 mm. at 640°C (m.p. 680°C). Vacuum sublimation at 220°C results in partial decomposition. Equilibrium H<sub>2</sub> pressure for NaH is 8.0 mm. at 300°C; for KH, 7.3 mm. at 300°C; for RbH, about 100 mm. at 370°C; for CsH, 0.3 mm. at 200°C and 27.8 mm. at 300°C. d. (x-ray) for LiH to CsH: 0.77; 1.36; 1.43; 2.59; 3.41. Crystal structure B1 type.

**REFERENCES:** 

- E. Zintl and A. Harder, Z. phys. Chem. (B) <u>14</u>, 265 (1931); see
   E. Zintl, A. Harder and S. Neumayr, Z. phys. Chem. (A) <u>154</u>, 92 (1931).
- K. Ziegler, H. G. Gellert, H. Martin, K. Nagel and J. Schneider, Liebigs Ann. Chem. 589, 91 (1954).
- P. Albert and I. Mahé, Bull. Soc. Chim. France [5] 17, 1165 (1950).
- G. F. Hüttig and A. Krajewski, Z. anorg. allg. Chem. <u>141</u>, 133 (1924).
- F. Ephraim and E. Michel, Helv. Chim. Acta 4, 724, 900 (1921).

### Alkali Metal Oxides

#### Li<sub>2</sub>O, Na<sub>2</sub>O, K<sub>2</sub>O, Rb<sub>2</sub>O, Cs<sub>2</sub>O

LITHIUM OXIDE

$$\begin{array}{c} \text{Li}_2\text{CO}_3 = \text{Li}_2\text{O} + \text{CO}_2 \\ 73.9 & 29.9 & 44.0 \end{array}$$

Zintl, Harder and Dauth prepared  $\text{Li}_2\text{O}$  by thermal decomposition of pure  $\text{Li}_2\text{CO}_3$ . Pure lithium carbonate (for purification see p. 987) is decomposed in a Pt boat set inside a porcelain tube which is connected to a mercury diffusion pump. Gas evolution ceases after heating for 50 hours at 700°C, as indicated by a McLeod gauge. The boat then contains pure white oxide, the composition of which can be checked by titration of samples. For preparation of Li<sub>2</sub>O from Li<sub>2</sub>O<sub>2</sub>, see under Li<sub>2</sub>O<sub>2</sub> (p. 979).

PROPERTIES:

More stable than the other alkali oxides; begins to sublime below 1000°C. M.p. above 1625°C. Reaction with water less vigorous. d. (x-ray) 2.00. C1 structure type, (Fluorite).

REFERENCES:

E. Zintl, A. Harder and B. Dauth, Z. Elektrochem. 40, 588 (1934).

SODIUM OXIDE

I.

 $5 \text{ NaN}_3 + \text{ NaNO}_3 = 3 \text{ Na}_2\text{O} + 8 \text{ N}_2$ 325.1 85.0 186.0 224.1

Direct oxidation of Na cannot be used to prepare pure  $Na_2O$ , since the simultaneously formed peroxide is reduced only with great difficulty to  $Na_2O$  by the excess Na. Zintl and Von Baumbach give the following directions for the preparation of pure  $Na_2O$ .

A nickel boat s (Fig. 269), lined with pure sodium azide, is charged with a finely powdered mixture of NaNO<sub>3</sub> (purified by recrystallization and dried at 200°C) and about 5.5 times (by weight) as much NaN<sub>3</sub>. The boat is placed in a Pyrex tube a and slowly heated by means of an electric furnace to 200°C in vacuum (Hg diffusion pump). When the reagents have thus been freed of moisture, the temperature is slowly raised to 270-290°C. (An explosion may occur on too rapid heating). The azide decomposes, imparting a dark gray color to the salt mixture. The stopcock



Fig. 269. Preparation of sodium oxide; a) Pyrextube; r) nickel tube; s) nickel boat.

leading to the pump is occasionally closed so that the course of the reaction can be followed by means of manometer m. The brown color (nitride formation) disappears with rising temperature. When N<sub>2</sub> evolution decreases, the temperature is raised to  $350^{\circ}$ C. Finally, the excess Na is distilled off in vacuum at  $350-400^{\circ}$ C. The Ni tube r protects the Pyrex from corrosion by Na vapor. Before cooling, the Na mirror is removed from the lower half of tube b by heating with a flame. Thus, no sodium should adhere to the boat as it is withdrawn from the tube. While the Na is being removed from the tube, the furnace must be left on to prevent Na condensation on top of the Na<sub>2</sub>O in the boat. The boat containing the preparation is then pulled out through ground joint c, making sure that air is completely excluded. It takes 3 to 5 hours to prepare 0.3-0.5 g. of Na<sub>2</sub>O. The yield is quantitative, based on NaNO<sub>3</sub>.

II. 
$$N_{a}OH + N_{a} = N_{a_{2}}O + \frac{1}{2}H_{2}$$
  
40.0 23.0 62.0 1.0

This method, according to Klemenc, Ofner and Wirth, enables the preparation of up to 8 g. of  $Na_2O$  in a single batch. A nickel crucible is placed in a Pyrex tube, closed off at the bottom and connected at the top to a vacuum line and a long-stem manometer. A mixture of NaOH granules and small Na pieces is used. Because NaOH generally contains a few percent of water, a correspondingly larger quantity of Na must be weighed out. Reaction begins at  $300-320^{\circ}C$  and the H<sub>2</sub> formed is continuously pumped out, maintaining the pressure at 30-40 mm. Finally, the slight excess of Na is distilled off under high vacuum. The product should be pure white. Its average composition is 96%  $Na_2O$ , 2% NaOH, 2%  $Na_2CO_3$ . It is best stored under anhydrous benzene.

### PROPERTIES:

Fine, white powder. d. (x-ray) 2.39.  $CaF_2$  lattice (C1). For corrosion resistance of various crucible materials to molten  $Na_2O$  and  $Na_2O_2$ , see E. G. Bunzel and E. J. Kohlmeyer, Z. anorg. Chem. 54, 4 (1947). According to this reference,  $Al_2O_3$  and NiO are relatively corrosion-resistant pure oxides.

### **REFERENCES:**

- I. E. Zintland H. H. von Baumbach, Z. anorg. allg. Chem. <u>198</u>, 88 (1931).
- II. A. Klemenc, G. Ofner and H. Wirth, Z. anorg. allg. Chem. <u>265</u>, 221 (1951).

POTASSIUM OXIDE

$$\begin{array}{rrr} 2 \mathrm{K} + \frac{1}{2} \mathrm{O}_2 = \mathrm{K}_2 \mathrm{O} \\ 78.2 & 16.0 & 94.2 \end{array}$$

According to Zintl, Harder and Dauth,  $K_2O$  is best prepared in the following way. Metallic K is repeatedly remelted under xylene to remove high-boiling oils. It is next degassed by double distillation in a Pyrex tube connected to a vacuum pump and finally filtered under high vacuum through a capillary into a Vycor boat. Thoroughly purified dry air is admitted in small quantities to the mildly heated bright metal. The oxide as finely divided particles absorbs the unreacted metal, liquefied by the heat of reaction, like a sponge. A mass having a mosslike structure and a metallic sheen is obtained. Only a portion of the K is thus oxidized at one time (to prevent peroxide formation). The excess metal is distilled off in high vacuum at  $350^{\circ}C$ . The  $K_2O$  which remains in the boat is free of peroxide and is at least 99.5% pure.

**PROPERTIES:** 

Loose powder, yellow when hot but white at room temperature. Deliquescent in air. Reacts vigorously with  $H_2O$ . At  $350-400^{\circ}C$  disproportions into  $K_2O_2 + K$ . d. (x-ray) 2.33. CaF<sub>2</sub> lattice (C1).

**REFERENCES:** 

E. Zintl, A. Harder and B. Dauth, Z. Elektrochem. <u>40</u>, 588 (1934). E. Rengade, Comptes Rendus Hebd. Seances Acad. Sci <u>144</u>, 754 (1907).

RUBIDIUM OXIDE

I.

$$\begin{array}{rrrr} 2 \ \mathrm{Rb} \ + \ {}^{1}\!/_{2} \ \mathrm{O}_{2} \ = \ \mathrm{Rb}_{2} \mathrm{O} \\ 171.0 & 16.0 & 187.0 \end{array}$$

According to Helms and Klem,  $Rb_2O$  is prepared by the same method as  $K_2O$ . The metal is reacted in a glass apparatus with a quantity of  $O_2$  insufficient to prevent peroxide formation. The excess metal is then distilled off.

II. 
$$2 \text{ Rb} + \text{HgO} = \text{Rb}_2\text{O} + \text{Hg}$$
  
171.0 216.6 187.0 200.6

Another way to prepare  $Rb_2O$  is to react the metal with less than the stoichiometric quantity of HgO. The metal vapor must be distilled into the HgO in small portions to reduce the intensity of the reaction. After the preparation has been held 12 hours at 200°C, the excess Rb and the Hg formed in the reaction are distilled off in high vacuum at about  $200^{\circ}$ C. The Rb<sub>2</sub>O remains.

#### PROPERTIES:

Colorless powder at room temperature, yellow when heated. Decomposed by light, turning dark. Vigorous reaction with  $H_2O$ . Above 400°C disproportionates into  $Rb_2O_2 + Rb$ . d. (x-ray) 3.72. CaF<sub>2</sub>, lattice (C1).

**REFERENCES:** 

A. Helms and W. Klemm, Z. anorg. allg. Chem. 242, 33 (1939). E. Rengade, Compt. Rend. Hebd. Séances Acad. Sci. 144, 754 (1907).

#### CESIUM OXIDE

$$\begin{array}{rrrr} 2 \ C_{S} \ + \ {}^{1/_{2}} \ O_{2} \ = \ C_{S_{2}} O \\ 265.8 \ 16.0 \ 281.8 \end{array}$$

According to Helms and Klemm, the reaction of Cs with HgO cannot be used to prepare pure  $Cs_2O$ . Instead, in a procedure

identical to that described for  $K_2O$  and  $Rb_2O$ , metallic Cs is incompletely oxidized and the excess metal is distilled away in high vacuum at 200°C. Care must be taken that no cesium diffuses back into the Cs<sub>2</sub>O during cooling.

Brauer has described an apparatus for the preparation of  $Cs_2O$ , shown in Fig. 270. Boat  $\alpha$  is shaped like a slipper so that it is able to contain liquid Cs in both vertical and horizontal positions. A nitrogen stream is passed through the horizontal reactor tube, which is open at the ground glass joint. A weighed

ampoule, fastened to a wire and filled with pure Cs, is inserted into the reactor tube as shown. Prior to its insertion, the pointed end of the ampoule is scratched a little and tapped with a hot glass rod, so that it may be broken off later on. After insertion, the point is broken off, with the ampoule at H and under a N<sub>2</sub> blanket. The broken off piece of glass is removed. The tube is then set vertically, the ampoule suspended inside, and a ground cap put on. Next the tube is evacuated and the Cs is transferred into the boat by careful melting. After cooling, the empty ampoule is pulled out (still under a stream of N<sub>2</sub>) and reweighed. Electrolytic O<sub>2</sub> is measured out by means of a glass burette (using Hg as the sealing



Fig. 270. Preparation of cesium oxide.

liquid), the amount introduced being insufficient for complete oxidation of the Cs to Cs<sub>2</sub>O. The oxygen is added to the reaction tube through a short capillary tube, which had been thoroughly evacuated during the evacuation of the main reactor tube. On its way from the burette to the reactor, the oxygen passes through a small U tube, cooled to a low temperature in order to condense the Hg and  $H_2O$ . Before entering the burette, the  $O_2$  is purified by passing through a cotton filter, a layer of palladium asbestos heated to  $400^{\circ}C$ , and a low-temperature trap.

The oxidation begins as soon as  $O_2$  reaches the Cs. To avoid an excessively vigorous reaction, the tube is cooled and the  $O_2$  is admitted in small portions. The product develops a brown-black color, and then becomes liquid, but resolidifies as the oxidation proceeds. To keep it liquid, which ensures a more thorough reaction, the cooling is stopped. When conditions warrant it, some heat is applied, but care must be taken to prevent the Cs from evaporating. Finally, the excess Cs is slowly distilled off, with the tube in the horizontal position and the furnace over the tube. The Cs is allowed to condense in portion b of the tube, which is enclosed in a cardboard box filled with solid  $CO_2$ . After the bottom end of the tube is broken off at c, the boat containing the pure Cs<sub>2</sub>O is transferred to another vessel under a blanket of nitrogen.

PROPERTIES:

Orange, but dark in transmitted light. Habit: according to Helms and Klemm, matted needles; according to Brauer, soft laminae cleaved along the base. d. (x-ray) 4.68. Probably has a  $CdI_2$  lattice (C6). M.p. (dec.) about 490°C. Deliquescent in air. Vigorous reaction with water (ignites), less violent with alcohol.

**REFERENCES:** 

A. Helms and W. Klemm, Z. anorg. allg. Chem. <u>242</u>, 33 (1939). G. Brauer, Z. anorg. Chem. 255, 119 (1947).

E. Rengade, Compt. Rend, Hebd. Séances Acad. Sci. <u>144</u>, 753 (1907); Ann. Chim. Phys. [8] <u>11</u>, 384, 388 (1907).

# Lithium and Sodium Peroxides

Li<sub>2</sub>O<sub>2</sub> and Na<sub>2</sub>O<sub>2</sub>

### LITHIUM PEROXIDE

According to Pierron,  $\text{Li}_2O_2$  is prepared by treating a boiling, saturated solution of LiOH in 95% alcohol with the stoichiometric

quantity of Perhydrol. The solution is decanted from the precipitate. The latter is boiled twice (30 minutes each time) in 95% alcohol, separated, and then dried overnight in vacuum over  $P_2O_5$ . The product contains 99.4% Li<sub>2</sub>O<sub>2</sub>. Starting with 10 g. of LiOH, 5-6 g. of Li<sub>2</sub>O<sub>2</sub> is obtained. The Li in the alcoholic mother liquor can, however, be reused.

Very pure  $\text{Li}_2O_2$  can easily be obtained from  $\text{Li}_2O_2$ , which can initially contain  $\text{Li}_2O$  for this purpose, by heating in a glass tube at 300°C in vacuum. Pure  $\text{Li}_2O$  forms in theoretical amounts as a perfectly white powder.

PROPERTIES:

d. 2.14. Structure: probably similar to Hg(I) halides (D3<sub>1</sub>type).

SODIUM PEROXIDE

$$\begin{array}{r} 2\,Na \,+\,O_2 \,=\, Na_2O_2 \\ 46.0 \quad 32.0 \quad 78.0 \end{array}$$

In preparing  $Na_2O_2$  it is advantageous to oxidize metallic Na to  $Na_2O$  in an atmosphere containing less  $O_2$  than air. The  $Na_2O$  is then oxidized completely to  $Na_2O_2$  in an atmosphere containing more  $O_2$  than air, at temperatures ranging from 200 to  $350^{\circ}C$ . If the  $Na_2O$  is ground before the second oxidation step (to facilitate the reaction), air moisture must be excluded during grinding and reaction.

PROPERTIES:

Yellowish. M.p.  $460^{\circ}$ C (does not decompose). Forms  $H_2O_2$  with  $H_2O$ . At red heat decomposes, evolving  $O_2$ . d. 2.47. Structure probably similar to Hg(I) halides (D3<sub>1</sub> type).

REFERENCES:

P. Pierron, Bull. Soc. Chim. France (5) <u>6</u>, 235 (1939).
F. Fehér, Angew. Chem. <u>51</u>, 497 (1938).
Roessler and Hasslacher Chem. Co., British patent 264,724 (1926).

# Alkali Dioxides

NaO<sub>2</sub>

$$\begin{array}{rrr} Na_2O_2 \ + \ O_2 \ = \ 2 \ NaO_2 \\ 78.0 \ & 32.0 \ & 110.0 \end{array}$$

In the method of Stephanou, Schlechter, Argersinger and Kleinberg, 92% pure Na<sub>2</sub>O is prepared by the action of O<sub>2</sub> on Na<sub>2</sub>O<sub>2</sub> at ~500°C and 300 atm. A high-grade steel bomb (180-ml. capacity) provided with a thermocouple is connected through a needle valve to a vacuum pump and an oxygen cylinder. A weighed amount (about 10 g.) of very pure Na<sub>2</sub>O<sub>2</sub>, free from carbonate, is placed in an open Pyrex vessel and introduced into the bomb. After evacuating for several hours at 1-3 mm., enough O<sub>2</sub> is pumped to raise the initial pressure in the bomb to about 300 atm. when the latter is heated to 500°C. The valve is then closed. When no further pressure drop is noted, indicating that the reaction has been completed (in 100 hours, at the most), the bomb is left to cool. The fused product is readily powdered in the absence of moisture.

KO<sub>2</sub>, RbO<sub>2</sub> and CsO<sub>2</sub>

K (Rb, Cs) + 
$$O_2 = KO_2$$
 (Rb $O_{2}$ , Cs $O_2$ )  
39.1 (85.5; 132.9) 32.0 71.1 (117.5; 164.9)

According to Klemm and Sodomann, as well as Helms and Klemm, the best method for preparing  $KO_2$ ,  $RbO_2$  and  $CsO_2$  is by oxidation of the elements, dissolved in liquid  $NH_3$ , with  $O_2$  at -30 to -50°C. Intermediate products appear first. Their color is light yellow when fresh, then dark. Finally, yellow dioxides are formed. To prevent an explosion, which is common in this reaction, Lux



Fig. 271. Reaction vessel for preparing alkali dioxides by oxidation with  $O_2$  of alkali metals dissolved in liquid NH<sub>3</sub>.

and Kuhn suggest the following procedure, using  $K_2O$  as an example.

Ammonia gas, predried with KOH, is first passed for 2-3 hours through reaction vessel q (Fig. 271) until all of the O<sub>2</sub> is displaced. An acetone-Dry Ice bath (about  $-50^{\circ}$ C) is then placed under the vessel. The circular copper channel r is fitted tightly around the upper third of g. It is filled with an acetone-Dry Ice mixture at -70 to -80°C. As soon as the lower third of reaction vessel q is filled with liquid NH<sub>3</sub>, a piece of potassium is placed in the adapter a and carefully melted by electrical heating. The reasonably pure metal trickles through capillary b (diameter about 1 mm.) into the liquid NH<sub>3</sub>. Spattered K, which might easily cause an explosion, is completely washed off the walls by refluxing  $NH_3$ , using channel r as a reflux condenser.

After all the K is dissolved, adapter c is quickly replaced with a plain cap d. The stopcock is then quickly switched over to

admit  $O_2$ , prepurified with soda-lime,  $CaCl_2$  and  $P_2O_5$ . A white precipitate of  $K_2O_2$  appears first, followed by a brick-red intermediate product (possibly  $K_2O_3$ ). The latter is finally converted to yellow  $KO_2$ . After about four hours, the coolant is removed and all remaining NH<sub>3</sub> is evaporated in a stream of  $O_2$ . Finally, the freshly formed  $KO_2$  is heated, in an oil-pump vacuum, with a nonluminous flame and transferred to a suspended collector tube. The oxygen content of the preparation is close to theoretical.

#### PROPERTIES:

Yellow substances, decomposed by  $H_2O$  with evolution of  $O_2$ . The structure of  $NaO_2$  is similar to that of NaCl, with  $O_2$  replacing the Cl<sup>-</sup> ions. Structures of  $KO_2$ ,  $RbO_2$  and  $CsO_2$ :  $CaC_2$  lattice (C11).

	d.(x-ray)	M.p.°C
NaO <sub>2</sub>	2.21	_
KO <sub>2</sub>	2.14	380
RbŌ,	3.06	412
CsO,	3.80	432

#### **REFERENCES:**

- NaO<sub>2</sub>: S. E. Stephanou, W. H. Schlechter, W. I. Argersinger and J. Kleinberg. J. Amer. Chem. Soc. <u>71</u>, 1819 (1949); D. H. Templeton and C. H. Dauben, J. Amer. Chem. Soc. <u>72</u>, 2251 (1950);
- KO<sub>2</sub>, RbO<sub>2</sub>, CsO<sub>2</sub>: W. Klemm and H. Sodomann, Z. anorg. allg. Chem. <u>225</u>, 273 (1935); A. Helms and W. Klemm, Z. anorg. allg. Chem. <u>241</u>, 97 (1939); H. Lux and R. Kuhn, 1957; A. Joannis, Compt. Rend. Hebd. Séances Acad. Sci. <u>116</u>, 1370 (1893); C. A. Kraus and E. F. Parmenter, J. Amer. Chem. Soc. <u>56</u>, 2384 (1934).

### Lithium Hydroxide

#### LiOH · H<sub>2</sub>O, LiOH

To prepare lithium hydroxide by the method of Barnes, equivalent amounts of aqueous solutions of  $\text{Li}_2\text{SO}_4$  and  $\text{Ba}(\text{OH})_2$  are

reacted and the filtrate is concentrated in vacuum on a Pt dish. The product solution tends to become supersaturated. It takes three weeks before large needle-shaped crystals of the monohydrate LiOH  $\cdot$  H<sub>2</sub>O are formed.

According to De Forcrand, the powdered monohydrate is converted to LiOH on drying for several days over  $P_2O_5$  in vacuum. The dehydration can also be accomplished by slow heating of the monohydrate of 140°C in a silver boat, white in a stream of pure  $H_2$ . If the temperature is raised too rapidly, the preparation melts at 445°C and converts to a hydrate 8LiOH  $\cdot$  H<sub>2</sub>O, which can be dehydrated only with difficulty. At 660-780°C the compound loses all its water and Li<sub>2</sub>O remains as a residue.

**PROPERTIES:** 

LiOH: white, translucent, less hygroscopic than NaOH. M.p. 462°C. d. 1.46 B10 structure type.

LiOH  $\cdot$  H<sub>2</sub>O: a saturated solution at 25°C contains 7.15% Li<sub>2</sub>O. d. 1.51 B36 structure type.

**REFERENCES:** 

E. Barnes, J. Chem. Soc. (London) 1931, 2605.

De Forcrand, Comptes Rendus Hebd. Séances Acad. Sci. 146, 803 (1908).

## **Rubidium and Cesium Hydroxides**

### **RbOH**, CsOH

 $\begin{array}{rl} Rb_2SO_4 \ (Cs_2SO_4) \ + \ Ba(OH)_2 \ = \ 2 \ RbOH \ (CsOH) \ + \ BaSO_4 \\ 267.0 \ \ (361.9) \ \ 171.4 \ \ 205.0 \ \ (299.8) \ \ 233.4 \end{array}$ 

To prepare RbOH and CsOH according to Barnes, aqueous solutions of  $Rb_2SO_4$  (or  $Cs_2SO_4$ ) and  $Ba(OH)_2$  are reacted in equivalent proportions. The filtrate is concentrated in a Pt dish in vacuum over solid KOH and heated slowly at  $300^{\circ}C$  in a silver boat in a stream of  $CO_2$  free hydrogen.

According to Von Hevesy, the last persistently adhering traces of water are removed by bubbling through purified and dried  $N_2$  through the melt with the aid of a thin Ag tube.

According to Winslow, Liebhafsky and Smith, pure RbOH and CsOH are prepared by electrolysis of the chlorides, using an amalgam procedure carried out in a multicell apparatus.

The hydroxides can be worked up to the azides, chromates and iodides.
PROPERTIES:

White, crystalline, very hygroscopic substances. RbOH and CsOH: m.p.  $301^{\circ}$ C and  $272.3^{\circ}$ C, Enantiotropic transformation points  $245^{\circ}$ C and  $223^{\circ}$ C. Saturated aqueous solutions at  $30^{\circ}$ C contain 65.56% Rb<sub>2</sub>O and 70.63% Cs<sub>2</sub>O. Soluble in alcohol. d. 3.20 and 3.675.

REFERENCES:

- G. von Hevesy, Z. phys. Chem. 73, 667 (1910).
- A. F. Winslow, H. A. Liebhafsky and H. M. Smith, J. Phys. Colloid. Chem. <u>51</u>, 967.

#### Lithium Nitride

#### Li<sub>3</sub>N

 $\begin{array}{l} 3\,\text{Li}\,+\,{}^{1}\!/_{2}\,\text{N}_{2}\,=\,\text{Li}_{3}\text{N}\\ 20.8\,\,\,\,14.0\,\,\,\,34.8 \end{array}$ 

The method of Zintl and Brauer uses pieces of pure Li milled clean under an Ar blanket. These pieces are then transferred in the absence of air to a vessel described by Zintl and Woltersdorf (Fig. 272) and nitrided. The crucible t is made of zirconium dioxide coated with fused lithium fluoride (m.p.  $840^{\circ}$ C). In contrast to most ceramic and metal vessels, these crucibles are completely resistant to liquid Li up to  $800^{\circ}$ C, thus allowing the preparation of pure Li compounds from the metal. Crucible t is set inside an iron protec-

tive crucible e, which is placed in a ceramic tube r. The top adapter g is cemented to r with sealing wax. A lead cooling coil is wrapped around the cemented area. Stopcock h may be used both as a vacuum connection and a gas inlet. A peephole f cemented onto g, permits observation of the material in the crucible.

Large crystallites are more likely to be obtained if the nitriding is started at  $400^{\circ}$ C, the temperature is gradually raised to  $800^{\circ}$ C, and the thoroughly purified and dried N<sub>2</sub> is diluted with about 20 vol. % Ar.

Fig. 272. Preparation of lithium nitride: t) crucible made of  $ZrO_2$  coated with LiF; e) iron protective crucible; r) ceramic outer tube; g) glass adapter with a glass peephole f and stopcock h.



PROPERTIES:

Ruby-red, translucent crystallites, decomposed by moisture with evolution of  $NH_3$ . d. (x-ray) 1.28. Hexagonal.

REFERENCES:

- E. Zintl and G. Brauer, Z. Elektrochem. <u>41</u>, 102 (1935).
- E. Zintl and G. Woltersdorf, Z. Elektrochem. 41, 876 (1935).

#### Phosphides, Arsenides, Antimonides and Bismuthides

	+ P 31.0	+ As 74.9	+ Sb 121.8	+ Bi 209.0
3 Li 20.8 3 Na 69.0 3 K 117.3	Li <sub>3</sub> P 50.8 Na <sub>3</sub> P 99.0	Li <sub>3</sub> As 95.7 Na <sub>3</sub> As 143.9 K <sub>3</sub> As 192.2	Li <sub>3</sub> Sb 142.6 Na <sub>3</sub> Sb 190.7 K <sub>3</sub> Sb 239.0	Li <sub>3</sub> Bi 229.8 Na <sub>3</sub> Bi 278.0 K <sub>3</sub> Bi 326.3

of Alkali Metals from the Elements

According to Brauer and Zintl, the phosphides, arsenides, antimonides and bismuthides of Li, Na and K are prepared as follows.

Preparation of starting materials: Alkali metals stripped of crusts under light benzine, then freed of the latter in high vacuum; red phosphorus prepared from freshly distilled, completely dry yellow phosphorus by prolonged heating at  $275^{\circ}$ C; As, sublimed in a nitrogen stream; Sb and Bi of highest purity (see also the related sections of this book).

#### Li<sub>3</sub>P, Li<sub>3</sub>As, Li<sub>3</sub>Sb and Li<sub>3</sub>Bi

To prepare  $\text{Li}_3P$  (or  $\text{Li}_3As$ ), Li and red P (or Li and As) are fused in a crucible made of zirconium dioxide and lined with LiF (see under  $\text{Li}_3N$ ). The crucible is hermetically sealed in an iron crucible and heated to 680°C for  $\text{Li}_3P$  and to 800°C for  $\text{Li}_3As$ . Very pure red-brown  $\text{Li}_3P$ , or brown-black  $\text{Li}_3As$ , is obtained. It is transferred from the crucibles to glass vessels provided with ground stoppers and stored under Ar. The product should not come in contact with air at any time.

To obtain  $\text{Li}_3\text{As}$  and  $\text{Li}_3\text{Sb}$ , a suspension of freshly powdered As (or Sb) in anhydrous liquid NH<sub>3</sub> is prepared. A solution of Li in dried liquid ammonia is added, in small portions, to the suspension. The reaction begins on shaking. Its termination is indicated by the appearance of a blue color in the supernatant fluid, due to a slight excess of Li. The NH<sub>3</sub> is boiled out, with the last

traces removed by vacuum. The products are either the brown  $\text{Li}_3\text{As}$  powder (very air sensitive; must be stored under pure N<sub>2</sub> or Ar) or blue-gray  $\beta$ -Li<sub>3</sub>Sb powder. The particle size of the latter can be increased by heating at 650°C. Another modification,  $\alpha$ -Li<sub>3</sub>Sb, is obtained by quenching of the melt (m.p. between 1150 and 1300°C) under Ar in a thick-walled iron crucible with welded-on lid.

According to Zintl and Brauer,  $Li_3Bi$  is prepared by melting together stoichiometric quantities of Li and Bi in tall, narrow iron crucibles, followed by slow cooling. The crucibles are filled with Ar and tightly closed with welded-on iron stoppers. (See also Part III, section on intermetallic compounds.)

#### Na<sub>3</sub>P, Na<sub>3</sub>As, Na<sub>3</sub>Sb and Na<sub>3</sub>Bi

The phosphide forms when stoichiometric quantities of Na and red P are fused under a protective Ar blanket in a Tamman crucible. The latter is made of sintered corundum and is hermetically sealed in an outer iron crucible.

To prepare Na<sub>3</sub>As, Na vapor is passed over heated As. An iron boat filled with Na is placed inside an evacuated Vycor tube, together with another boat made of sintered corundum and filled with As powder. Using small movable electric tubular heaters, the Na is vaporized at  $350-450^{\circ}$ C and the vapor is passed over the As, preheated to  $180-200^{\circ}$ C. Conversion to brown-violet Na<sub>3</sub>As is complete. The excess Na is removed from the apparatus on heating to  $450^{\circ}$ C in high vacuum.

Both Na<sub>3</sub>Sb and Na<sub>3</sub>Bi are obtained from stoichiometric quantities of the respective reactants which are melted together in tall, narrow crucibles (filled with Ar and tightly closed with welded-on iron stoppers) at temperatures exceeding the melting point ( $856^{\circ}C$ and  $775^{\circ}C$ , respectively) by 50-100C°. (See the section on Intermetallic Compounds, Part III.) After slow cooling, the crucibles are cut open under a protective gas blanket. The products are brittle, bluish or violet-gray, and have a slight metallic sheen.

#### K<sub>3</sub>As, K<sub>3</sub>Sb and K<sub>3</sub>Bi

The arsenide is obtained in a manner analogous to  $Na_3P$  at  $800^{\circ}C$ . It is a very brittle substance with a greenish metallic luster.

Both K<sub>3</sub>Sb and  $\alpha$ -K<sub>3</sub>Bi are prepared in the same manner as the corresponding Na compounds (m.p. 812°C for K<sub>3</sub>Sb and 671°C for K<sub>3</sub>Bi). Both substances are very brittle, with a green-yellow sheen resembling that of fuchsin crystals ( $\beta$ -K<sub>3</sub>Bi is stable only above 280°C).

**PROPERTIES:** 

 $\beta$ -Li<sub>3</sub>Sb and Li<sub>3</sub>Bi crystallize in DO<sub>3</sub> structure type; all the others, in type DO<sub>15</sub>. D. (x-ray): Li<sub>3</sub>P 1.43; Li<sub>3</sub>As 2.42;  $\alpha$ -Li<sub>3</sub>Sb 2.96;  $\beta$ -Li<sub>3</sub>Sb 3.29; Li<sub>3</sub>Bi 5.03; Na<sub>3</sub>P 1.74; Na<sub>3</sub>As 2.36; Na<sub>3</sub>Sb 2.67; Na<sub>3</sub>Bi 3.70; K<sub>3</sub>As 2.14; K<sub>3</sub>Sb 2.35; K<sub>3</sub>Bi 2.98.

**REFERENCES:** 

G. Brauer and E. Zintl, Z. phys. Chem. (B) <u>37</u>, 323 (1937). E. Zintl and G. Brauer, Z. Elektrochem. 41, 297 (1935).

#### Sodium and Lithium Carbides

 $Na_2C_2$ ,  $Li_2C_2$ 

 $2 \text{ Na (Li)} + 2 \text{ } C_2\text{H}_2 = 2 \text{ NaHC}_2 \text{ (LiHC}_2) + \text{H}_2$  46.0 (13.9) 52.1 96.1 (63.9) 2.0  $2 \text{ NaHC}_2 \text{ (LiHC}_2) = \text{ Na}_2\text{C}_2 \text{ (Li}_2\text{C}_2) + \text{ C}_2\text{H}_2$  96.1 (63.9) 70.0 (37.9) 26.0

In the method of Antropoff and Müller,  $Na_2C_2$  may be prepared by letting acetylene react with Na dissolved in liquid NH<sub>3</sub>. The sodium acetylide thus formed loses acetylene at 145°C in vacuum and is converted to  $Na_2C_2$ . The latter is obtained as a pure white product, containing 1.3-3.0% undecomposed acetylide.

Lithium carbide, obtained in a similar way, invariably contains 20% acetylide, even when heated in vacuum to the decomposition temperature of the carbide (about  $300^{\circ}$ C).

PROPERTIES:

 $Na_2C_2$  is very hygroscopic. d. 1.575.  $Li_2C_2$  is white and crystalline; decomposed by water. d. 1.65.

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#### Alkali Metal Carbonates of Highest Purity

To purify commercial  $Li_2CO_3$  (as well as lithium carbonate obtained from minerals), it is dissolved in acetic acid, following

the method of Zintl, Harder and Dauth; the Ca is precipitated with ammonium oxalate and the Mg with  $Ba(OH)_2$  solution. The barium is removed from the filtrate by precipitation with sulfuric acid. The solution, separated from the precipitate, is evaporated to dryness and the residue mildly calcined to remove ammonium salts. It is then dissolved in hydrochloric acid and treated with distilled ammonium carbonate to precipitate pure  $Li_2CO_3$ .

Pure  $Na_2CO_3$ ,  $K_2CO_3$ ,  $Rb_2CO_3$  and  $Cs_2CO_3$  are prepared by the method of Suhrmann and Clusius, starting with pure alkali chlorides. These salts are converted to the nitrates by treatment with excess nitric acid in steamed-out Pyrex vessels. After removal of the chlorine, the nitrates are treated in a Pt dish with a fourfold amount of crystallized oxalic acid and are finally calcined to the pure carbonates. No foreign metals are used in this method; hence no impurities can be introduced.

#### PROPERTIES:

 $Li_2CO_3$ : formula weight 73.9. Colorless monoclinic crystals. M.p. 735°C, b.p. 1200°C. Solubility at 0°C, 1.54; at 20°C, 1.33; and at 100°C, 0.73 g./100 g. of H<sub>2</sub>O. d. 2.1.

Na<sub>2</sub>CO<sub>3</sub>: formula weight 106.0. Transition point 450°C. m.p. 860°C. Solubility at 20°C (solid phase Na<sub>2</sub>CO<sub>3</sub>  $\cdot$  10H<sub>2</sub>O), 21.58 g.; at 35°C (solid phase Na<sub>2</sub>CO<sub>3</sub>  $\cdot$  7H<sub>2</sub>O), 49.25 g.; and at 75°C (solid phase Na<sub>2</sub>CO<sub>3</sub>  $\cdot$  H<sub>2</sub>O), 45.88 g./100 g. of H<sub>2</sub>O. CO<sub>2</sub> pressure 1 mm. at 700°C. d. 2.53.

 $K_2CO_3$ : formula weight 138.2. Transition point 140°C. m.p. 891°C. Hygroscopic. Solubility at 0°C, 105 g.; at 20°C, 112 g.; and at 100°C, 156 g./100 g. of H<sub>2</sub>O. CO<sub>2</sub> pressure 1.2 mm. at 950°C. d. 2.43.

 $Rb_2CO_3$ : formula weight 231.0. m.p. 837°C. On melting, attacks Pt. Very hygroscopic. Solubility in water at 20°C, 69.01 g./100 g. of saturated solution. Solubility in absolute alcohol at 19°C, 0.74 g./100 g. of alcohol.  $CO_2$  pressure 2 mm. at 740°C; 10 mm. at 900°C.

 $Cs_2CO_3$ : formula weight 325.8. Melts at red heat. Deliquescent in air. Solubility in H<sub>2</sub>O: at 20°C, the saturated solution contains 72.34%  $Cs_2CO_3$ . Solubility in absolute alcohol at 19°C, 11.1 g./100 g. of alcohol; at the b.p. of alcohol, 20.1 g.  $CO_2$  pressure 2 mm. at 610°C; 44 mm. at 1000°C.

#### **REFERENCES:**

- E. Zintl, A. Harder and B. Dauth, Z. Elektrochem. 40, 588 (1934).
- R. Suhrmann and K. Clusius, Z. anorg. allg. Chem. <u>152</u>, 52 (1926).

#### Silicides and Germanides of Alkali Metals

#### from the Elements

NaSi, KSi, RbSi, CsSi, NaGe, KGe, RbGe, CsGe

		+ Si	28,1	+ Ge	72,6
Na	23.0	NaSi	51.1	NaGe	95.6
K	39.1	KSi	67.2	KGe	111.7
Rb	85.5	RbSi	113.5	RbGe	158.1
Cs	132,9	CsSi	161.0	CsGe	205.5

The alkali silicides and germanides are very moisture sensitive. In the method of Hohmann the synthesis is carried out in corundum crucibles set in vacuum-tight, Ar-filled iron bombs (Fig. 273, A and B). The corundum crucible i is thoroughly ignited before being charged with a small amount (a fraction of a gram) of Si (or Ge) which has been finely ground in an agate mortar. Crucible i is then placed in small bomb b, and the latter set on the stem of a Pyrex device (Fig. 274) designed for transfer of alkali metals. The apparatus is carefully evacuated through c (so as not to lose any Si or Ge through dusting), and Ar gas is let in through the same route and escapes through e. Next, ampoule f, containing the alkali metal and already opened at the bottom, is introduced through e and allowed to slide down into the slightly tilted apparatus. Ground cap e is then replaced and the apparatus is evacuated at once. The area B is then thoroughly heated with a gas burner. The alkali metal (present in a three- or fourfold excess) is caused to flow into the crucible by heating area f with a flame. After cooling, Ar is reintroduced and the bomb removed in a strong Ar stream through the bottom by lifting cap a. The bomb is immediately tightly closed with cone l, ring (or plate) gasket k and screw-cap h (see Fig. 273).

Following completion of the reaction (see below), the bomb is slowly cooled and opened by unscrewing the cap until 0.5 mm. of the thread is left, and then breaking off the cap is on a bench vise. The open bomb is quickly pushed into the distillation tube (Fig. 275), through which either Ar or  $N_2$  is flowing. After evacuating briefly, Ar is again admitted and the tube sealed at m. Finally, an electric furnace is slid over the tube and the excess of alkali metal is separated from the product by distillation in high vacuum and condensed in n. This may take several days. After breaking off the end of the tube at m, bomb b containing the corundum crucible is taken out in a stream of Ar. If necessary, the product can be ground in an apparatus proposed by Klemm and Dinkelacker, in the absence of air and moisture. (See section on Intermetallic Compounds.)



Fig. 273. Iron bombs used in the preparation of alkali metal silicides and germanides: A—bomb for preparation of Rb and Cs compounds; B—bomb for the Na and K compounds. b—Iron bomb; h—iron screw-on cap; i—corundum crucible; k—copper gasket; l—iron cone.

Fig. 274. Filling (transfer) apparatus for preparing alkali metal silicides and germanides: b-iron bomb; d-silicon or germanium; f-ampoule containing alkali metal.

Fig. 275. Separating the excess alkali metal from the silicide or germanide by distillation.

	Optimum reaction	Reaction	Distillat	ion
	temperatures	time	temperature	time
NaSi	700°C	1–2 days	280-300°C	4 days
KSi	650	4–5 days	240-250	several
RbSi CsSi NaGe	600 600 650-700	3–4 days 3–4 days 2 days	180-200 150-180	days
KGe, RbGe, CsGe	600	several days		

Supplementary Synthesis Data

#### PROPERTIES OF SILICIDES:

Very sensitive to moisture. Reactivity increases from NaSi to CsSi: NaSi self-igniting only as a loose powder; KSi self-igniting with detonation. All four silicides ignite explosively on contact with water or dilute acids. With dilute alkalis the reaction is milder.

NaSi: Long needles, metallic luster.

KSi: Hard, poorly crystallized substance with dark luster.

RbSi: Small dark crystals.

CsSi: Brittle, brass-colored compact mass; also single crystals. Decomposition temperatures in high vacuum; NaSi 420°C; KSi 360°C; RbSi and CsSi 350-360°C. Decomposition products: Na + Si, KSi<sub>8</sub>, RbSi<sub>8</sub> and CsSi<sub>8</sub>.

PROPERTIES OF GERMANIDES:

Sensitive to moisture, but less reactive than the silicides. Decompose in air into alkali hydroxide and brown germanium monohydride (GeH)<sub>x</sub>. Decompose rapidly, with occasional igniting in water, dilute acids or dilute alkalies.

NaGe: Small, well-formed needles, metallic luster.

KGe: Loose, dark-colored substance; no well-defined crystals. RbGe: Bronze-colored crystals.

CsGe: Crystals of a beautiful jet-black color.

Decomposition temperature in high vacuum: NaGe 480°C; KGe 400-420°C; RbGe and CsGe 390-410°C. Decomposition products: Na + Ge, KGe<sub>4</sub>, RbGe<sub>4</sub> and CsGe<sub>4</sub>.

Li<sub>2</sub>Si, Li<sub>4</sub>Si

 $2 \text{Li} (4 \text{Li}) + \text{Si} = \text{Li}_2 \text{Si} (\text{Li}_4 \text{Si})$ 13.9 (27.8) 28.1 41.9 (55.8)

In the method of Klemm and Struck,  $\text{Li}_2\text{Si}$  and  $\text{Li}_4\text{Si}$  are prepared in small tubular nickel crucibles. These are charged with weighed amounts of Si and Li in the absence of air and moisture and heated under argon in a Thermax steel bomb provided with a screw-on cap. The reaction temperature for  $\text{Li}_2\text{Si}$  is 530°C and for  $\text{Li}_4\text{Si}$ , 630°C. After cooling, the products can be ground, if needed, in an apparatus proposed by Klemm and Dinkelbacker. (See section on Alloys and Intermetallic Compounds.)

PROPERTIES:

Sensitive to moisture.  $Li_2Si$  is dark violet;  $Li_4Si$  is silvergray. **REFERENCES:** 

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- W. C. Johnson and A. C. Weatley, Z. anorg. allg. Chem. 216, 282 (1934).
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- W. Klemm and M. Struck, Z. anorg. allg. Chem. 278, 117 (1955).

A

Ag 279  $Ag(CF_3COO)$ 205AgCN 661240AgF 241 $AgF_2$ 239 $Ag_2F$ AgNCS 671 493, 514  $Ag_2N_2O_2$  $AgPO_2(NH_2)_2$ 582 $Ag_2SiO_3$ 705 831 AlAs  $AlAsO_4$ 831 772AIB<sub>2</sub> AlB<sub>12</sub> 772806, 813  $AlBr_3$  $AlBr_3 \cdot H_2S$ 819 $Al_4C_3$ 832 $Al(C_2H_5)_3$ 810  $Al(C_2H_5)_2Br$ 809  $Al(CH_3COO)_3$ 835  $\mathrm{Al}(\mathrm{C}_{2}\mathrm{H}_{5})_{2}\mathrm{Cl}\cdot\mathrm{O}(\mathrm{C}_{2}\mathrm{H}_{5})_{2}$ 811  $Al(C_2H_5)_2H = 811$  $Al(C_5H_7O_2)_3$  836  $\mathrm{Al}(\mathrm{C}_{2}\mathrm{H}_{5})_{3}\cdot\mathrm{O}(\mathrm{C}_{2}\mathrm{H}_{5})_{2}$ 811  $Al(CN)_3 \cdot O(C_2H_5)_2$ 834AlCl<sub>3</sub> 680, 805, 812 $Al_2Cl_3H_3$  808  $AlCl_3 \cdot 6H_2O$  815  $AlCl_3 \cdot NH_3$ 817  $AlCl_3 \cdot PCl_5$ 818  $AICI_3 \cdot SO_2 \quad 817$  $Al_2Cl_6 \cdot SOCl_2$ 818225 $AlF_3$ AlF<sub>3</sub> · 3H<sub>2</sub>O 225 $AlH_3 \cdot N(CH_3)_3$ 809  $AlH_3 \cdot 2N(CH_3)_3 = 809$ 

 $(AIH_3)_n \cdot xO(C_2H_5)_2$ 807  $AlI_3$ 814 $AlI_3 \cdot 6NH_3$ 819 AlN 827 829 Al(N<sub>3</sub>)<sub>3</sub> Al<sub>2</sub>O<sub>3</sub> 822Al(OCH<sub>3</sub>)<sub>3</sub> 833  $Al(OC_2H_5)_3$ 834, 835  $Al(OC_2H_4)_3 N$ 835 134 $Al(OD)_3$  $Al(OH)_3$ 676, 810, 820Alooh 820  $Al_2O_3 \cdot 2SO_2 \cdot H_2O$ 824 $Al_2O_3 \cdot 3SO_2 \cdot xH_2O$ 824 AlP 829 AlPO<sub>4</sub> 831 $Al_2S_3$ 134, 700, 823AlSb 831  $Al_2Se_3$ 825  $Al_2Te_3$ 826 Ar 82 As591  $AsBr_3$ 597  $AsCl_3$ 596 179, 197  $AsF_3$  $AsF_5$ 198 $AsH_3$ 593  $As_2H_4$ 594  $AsI_2$ 598  $AsI_3$ 597  $As_2O_3$ 600 601  $As_2O_5$  $As_2O_5 \cdot \frac{5}{3}H_2O$ 601  $As_2S_5$ 603  $As_4S_4$ 603  $As_2Zn_3$ 594

# B

B 770 B<sub>2</sub>Al 772 $B_{12}Al$ 772BAsO<sub>4</sub> 'voi PRro 770, 781 B(CH<sub>3</sub>)<sub>3</sub>, 798  $B(C_{2}H_{5})_{3}$ 799  $B(C_{n}H_{2n+1})_{3}$ 800  $BCl_3$ 780  $BCl_2(C_nH_{2n+1})$ 803  $BF_3$ 219 $BF_2(n-C_4H_9)$ 802  $BF_3 \cdot 2H_2O$ 784 $BF_3 \cdot NH_3$  785  $BF_3 \cdot O(C_2H_5)_2$ 786 $B_2H_6$ 773  $\begin{array}{c} \mathbf{BH}_3 \cdot \mathbf{N}(\mathbf{CH}_3)_3 \\ \mathbf{BI}_3 & 782 \end{array}$ 778 $BI_3$ BN 789  $B(N_{3})_{3}$ 476 $B_{3}N_{3}Cl_{3}H_{3}$  779  $B_3N_3H_6$ 779 787  $B_2O_3$ B(OCH<sub>3</sub>)<sub>3</sub> 797 B<sub>3</sub>O<sub>3</sub>(CH<sub>3</sub>)<sub>3</sub> 800  $B_{3}O_{3}(n-C_{4}H_{9})_{3}$ 801 800  $B(OH)_2CH_3$  $B(OH)_2(n-C_4H_9)$ 801  $BPO_4$ 796  $B_2S_3$ 788922 Ba  $BaBr_2$ 930  $Ba(BrO_3)_2 \cdot H_2O$ 316933 BaCO<sub>3</sub>  $BaCS_3$ 674BaCl<sub>2</sub> 930  $Ba(ClO_3)_2 \cdot H_2O$ 314 $Ba(ClO_4)_2 = 320$  $Ba(ClO_4)_2 \cdot 3H_2O$  320  $BaF_2$  234

 $BaGeF_6$  215 BaH, 929  $Ba_{3}H_{4}(IO_{6})_{2}$  326  $Ba(H_2PO_2)_2 \cdot H_2O 557$ BaH2P2O6 · 2H,O 562 BaI<sub>2</sub> 930  $Ba(N_3)_2$  942  $Ba(N_3)_2 \cdot H_2O$ 942 $Ba_3N_2^{-}940^{-}$ BaO 933 BaO<sub>2</sub> 937  $BaO_2 \cdot 8H_2O$  936  $Ba_3(PO_2S_2)_2 \cdot 8H_2O$ 572 BaS 938  $Ba(SO_3F)_2$  173  $BaS_2O_6 \cdot 2H_2O$ 397 BaSe 939  $BaSeO_4$  939 BaSi 947 BaSiO<sub>3</sub> 706  $BaSi_2O_5$  706 BaTe 940 Be 887  $\operatorname{BeBr}_2$ 891 BeC<sub>2</sub> 899, 900 Be<sub>2</sub>C 899  $Be(CH_3COO)_2$  901 BeCO<sub>3</sub> 893 BeCl<sub>2</sub> 889  $BeF_2$ 231BeI<sub>2</sub> 892 Be(N<sub>3</sub>)<sub>2</sub> 899  $Be_3N_2$  898 BeO 893  $Be_4O(CH_3COO)_6$  901  $Be_4O(C_2H_5COO)_6$  902 Be(OH)<sub>2</sub> 894 Be<sub>4</sub>O(HCOO)<sub>6</sub> 902 BeS 895 BeSe 897 BeTe 897 Bi 620  $BiBO_3 \cdot 2H_2O$ 627 BiBr<sub>3</sub> 623  $BiCl_2$ 622BiCl<sub>3</sub> 621

 $BiF_3$ 201 $\operatorname{BiF}_{5}$ 202  $BiI_3$ 624BiICl<sub>2</sub> 622  $Bi_2O_3$  620  $Bi_2O_4 \cdot aq = 629$ BiOBr 624 BiOCl 622 BiOI 625 BiONO<sub>2</sub> 626  $BiONO_2 \cdot 0.5H_2O$ 626  $BIONO_3$  620 BiPO<sub>4</sub> 626  $BiPO_4 \cdot 3H_2O$ 626 Br<sub>2</sub> 275 BrCN 665 156 $\operatorname{Br} \mathbf{F}_3$  $\operatorname{Br} F_5$ 158Br<sub>2</sub> • 8H<sub>2</sub>O 276 BrN<sub>3</sub> 476  $Br(NO_3)_3 = 328$  $BrO_2$  306 Br<sub>2</sub>O 307 [BrPy<sub>x</sub>]ClO<sub>4</sub> 328  $[BrPy_2]F 328$  $[BrPy_{X}]NO_{3}$ 328

# С

C 630  $C_{30}AlCl_4 \cdot 2AlCl_3$ 644  $C_8Br$  643 CClF<sub>3</sub> 205  $CCl_2F_2$  151, 205  $C_2Cl_2F_4$  205  $C_2 Cl_3 F_3 = 205$ C<sub>8</sub>Cs 635  $C_{24}Cs$ 635  $C_{36}Cs$ 635 C<sub>48</sub>Cs 635  $C_{12}Cs(NH_3)_2$ 637 CF 640 CF<sub>4</sub> 203, 207 C<sub>4</sub>F 641 f CF<sub>3</sub>COOAg 205

 $C_2HCl_3F_2$  205 CHF<sub>3</sub> 204  $(C_5H_5NH)_2PbCl_6$  750  $C_{24}HSO_4 \cdot 2H_2SO_4 642$ CIF<sub>3</sub> 205 C<sub>8</sub>K 635  $C_{24}K$ 635 C<sub>36</sub>K 636  $C_{48}K$ 636  $C_{12}K(NH_3)_2$  637  $C_{12}Li(NH_3)_2$ 637 $(CN)_2$  661 CNBr 665 CNCl 662CNI 666  $C_{12}Na(NH_3)_2$ 637 CO 645 CO<sub>2</sub> 647  $C_{3}O_{2}$  648 COBrF 210COCl<sub>2</sub> 650 COCIF 208 COF<sub>2</sub> 206, 210  $\operatorname{COIF}$ 211(CONH)<sub>3</sub> 668 COS 654 COSe 655 C<sub>8</sub>Rb 635 C<sub>24</sub>Rb 635 C<sub>36</sub>Rb 635 635 C<sub>48</sub>Rb  $C_{12}Rb(NH_3)_2$ 637 CS<sub>2</sub> 652  $C_{3}S_{2}$  653  $C_{3}S_{2}Br_{6}$  653 CSe<sub>2</sub> 656 Ca 922  $Ca(AlH_4)_2$  806 CaBr<sub>2</sub> 930 CaC<sub>2</sub> 943 CaCN<sub>2</sub> 946  $CaCO_3$ 931 CaCl<sub>2</sub> 930  $Ca(ClO_4)_2 = 320$  $Ca(ClO_4)_2 \cdot 4H_2O 320$  $CaF_2$  233 CaGe 948

CaH<sub>2</sub> 929  $CaI_2$  930  $Ca_{3}N_{2}$  940 CaO 931 CaO<sub>2</sub> 936 CaO<sub>2</sub> • 8H<sub>2</sub>O 936 Ca(OH)<sub>2</sub> 934  $Ca_{3}P_{2}$  942  $Ca_{10}(PO_4)_6(OH)_2 = 545$  $Ca_{2}PbO_{4}$  760 CaS 938 CaSe 939 CaSeO<sub>4</sub> 939 CaSi 946 CaSi<sub>2</sub> 946 CaTe 940 CdF<sub>2</sub> 243 CeF<sub>3</sub> 247  $CeF_4$  24  $Cl_2$  272 247 $(ClBNH)_3$  779 ClCN 662 ClF 153 ClF<sub>3</sub> 155 Cl<sub>2</sub> · 6H<sub>2</sub>O 274 $ClN_3$  476  $ClNH_2$  477 ClNO<sub>3</sub> 326  $ClN(SO_3K)_2$ 508 ClO<sub>2</sub> 301 Cl<sub>2</sub>O 299  $Cl_2O_6$  303 Cl<sub>2</sub>O<sub>7</sub> 304 ClO<sub>2</sub>F 165 $ClO_3F$  166  $ClO_4F$  167  $[ClPy_{x}]NO_{3}$ 328CoCl<sub>2</sub> 267 CoF<sub>2</sub> 267  $CoF_3$  $\mathbf{268}$  $CoSO_2 \cdot 3H_2O$  393  $CrF_2$  256 CrF<sub>3</sub> 257  $CrF_3 \cdot 3H_2O$  258  $CrF_4$  258  $CrO_{2}F_{2}$  258 Cs 958

 $CsAl(SO_4)_2 \cdot 12H_2O$ 956 CsBrCl<sub>2</sub> 294 CsC<sub>8</sub> 635  $CsC_{24}$ 635  $CsC_{36}$ 636 CsC<sub>48</sub> 636  $Cs_2CO_3$  987 CsCl 951, 955 CsGe 989 CsH 971  $CsIBr_2$  297 CsICl<sub>2</sub> 296 CsN<sub>3</sub> 476  $Cs(NH_3)_2C_{12}$ 637  $CsO_{2}$  981  $Cs_2O$ 974 CsOH 983 369  $Cs_2S_2$  $Cs_2S_3$ 369  $Cs_2S_5$ 369  $Cs_2S_6$  369  $Cs_2SeCl_6$  425 CsSi 989  $Cs_2TeCl_6$  444  $Cs(Tl_2Cl_9) = 874$ CuF<sub>2</sub> 238  $CuF_{2} \cdot 5H_{2}O \cdot 5HF$ 238 $Cu_2P_4O_{12} = 553$ D  $D_2 121$ DBr 131 DCl 129 DF 127

DH 126

DI 133

D<sub>2</sub>O 119

 $D_2S$  134

 $D_{3}PO_{3}$  132

 $D_{3}PO_{4}$  138

 $D_{2}SO_{4}$  135

EuF<sub>2</sub> 248

Ε

 $\begin{array}{rrrr} F_2 & 143 \\ F_2O & 163 \\ F_2O_2 & 162 \\ FSO_2NO & 186 \\ FeF_2 & 266 \\ FeF_3 & 266 \end{array}$ 

# G

F

Ga 837 GaAs 857  $GaBr_2$  846  $GaBr_3 = 845$ Ga(CH<sub>3</sub>)<sub>3</sub> 840  $Ga(CH_3)_3 \cdot N(C_2H_5)_3$ 841  $GaCl_2$  846  $GaCl_3$  843  $Ga(CIO_4)_3 \cdot 6H_2O$  839 GaF<sub>3</sub> 227  $Ga_2H_6$  840 Ga<sub>2</sub>H<sub>2</sub>(CH<sub>3</sub>)<sub>4</sub> 840 GaI<sub>3</sub> 846 GaN 855 Ga(N<sub>3</sub>)<sub>3</sub> 476  $Ga(NO_3)_3 = 856$ Ga<sub>2</sub>O 849 Ga<sub>2</sub>O<sub>3</sub> 848  $Ga(OH)_3$  847 GaO(OH) 847 GaP 857 GaS 851  $Ga_2S 852$  $Ga_2S_3 = 850$ GaSb 857 GaSe 854  $Ga_2Se 854$  $Ga_2Se_3 854$ GaTe 855  $Ga_2Te_3 855$ Ge 712  $GeBr_4$  718 Ge(CH<sub>3</sub>COO)<sub>4</sub> 726 GeCH<sub>3</sub>I<sub>3</sub> 722 GeCl<sub>2</sub> 716

HNCO 667, 668

 $GeCl_4$  707, 715 GeF<sub>4</sub> 215 GeH₄ 713Ge<sub>2</sub>H<sub>6</sub> 713 713Ge<sub>3</sub>H<sub>8</sub> GeHCl<sub>3</sub> 717, 721 720 GeI,  $GeI_4$ 719  $Ge_{3}N_{4}$  722 Ge(NH)<sub>2</sub> 723  $Ge_2N_3H$  723 GeO 711 GeO<sub>2</sub> 706  $Ge(OC_2H_5)_4$ 725GeS 723, 724 GeS<sub>2</sub> 723

## H

H<sub>2</sub> 111  $HAlBr_4 \cdot 20(C_2H_5)_2$ 817  $HAlCl_4 \cdot 20(C_2H_5)_2$ 816  $H_3AsO_4$  601 H<sub>3</sub>AsO<sub>4</sub>.0.5H<sub>2</sub>O 601  $H_7AsO_6$  601 HBF<sub>4</sub> 221  $H[BF_2(OH)_2] 784$  $(\text{HBNH})_3$  779 HBO<sub>2</sub> 791 282HBr HBrO<sub>3</sub> 315 HCN 658, 668  $H_2CS_3 = 674$ 280 HCl HClO 308  $HClO_3$  312 HClO<sub>4</sub> 318 126HD HF 145 HI 286 $HICl_4 \cdot 4H_2O$ 299 HIO<sub>3</sub> 316  $H_5IO_6$  322  $HIO_3 \cdot I_2O_5$ 307  $HN_3 472$ 

HNCS 669 HNO<sub>3</sub> 491  $H_2N_2O_2$  492  $H_{2}O$  117  $H_2O_2$  140  $H_3PO_2$ 555  $H_3PO_3$ 554 H<sub>3</sub>PO<sub>4</sub> 543 558 H<sub>4</sub>P<sub>2</sub>O<sub>6</sub>  $H_4P_2O_6 \cdot 2H_2O$ 559  $H_4P_2O_7$  546  $HPO_2Cl_2$  538  $HPO_2(NH_2)_2$ 582 $H_2PO_3NH_2$  579  $H_3PO_3S$  568  $H_{2}S$  344  $H_2S_2$ 350  $H_2S_3$ 350  $H_2S_4$ 353  $H_2S_5$ 353  $H_2S_6$ 353, 355  $H_2S_7$ 353, 355  $H_2S_8$ 353, 355  $H_2S_X$ 346 $H_2SO_5$  388  $H_2S_2O_8$ 389  $H_2S_XO_3$ 405  $H_2S_XO_6$ 405HSO<sub>3</sub>Cl 385 HSO3F 177  $HSO_3NH_2$  508  $HSbCl_{6} \cdot 4.5H_{2}O$  611 H<sub>2</sub>Se 418  $H_2SeO_3$ 430H₂SeO₄ 432 $H_2SiF_6$ 214 $H_{2}Si_{2}O_{3}$  694, 699  $H_2Si_2O_5$ 699  $H_4SiO_4$ 697  $H_2Te 438$ H<sub>2</sub>TeO<sub>3</sub> 449 H<sub>6</sub>TeO<sub>6</sub> 451  $H(TlCl_4) \cdot 3H_2O = 872$  $HTl(SO_4)_2 \cdot 4H_2O 882$  $H_2SnCl_6 \cdot 6H_2O$  730 He 82

 $\begin{array}{rrrr} Hg & 28 \\ Hg_2CO_3 & 243 \\ HgF_2 & 244 \\ Hg_2F_2 & 243 \\ HgO & 299 \\ HgSeF_4 & 180 \\ \end{array}$ 

# I

 $I_2 277$ IBr 291  $[I(C_5H_5N)_2]ClO_4$ 327ICN 666 ICl 290 ICl<sub>3</sub> 292 I(ClO<sub>4</sub>)<sub>3</sub> 330  $IF_5$ 159  $IF_7$ 160I(IO<sub>3</sub>)<sub>3</sub> 331I(NO<sub>3</sub>)<sub>3</sub> 329 $I_2O_4$  333  $I_2O_5$ 307  $I_4O_9$  331  $I_2O_5 \cdot HIO_3 \quad 307$  $(IO)_2 SO_4 \cdot H_2 O = 342$ [IPy<sub>X</sub>]ClO<sub>4</sub> 328[IPy<sub>2</sub>]F 328  $[IPy_X]NO_3$ 328 $I_2(SO_4)_3$ 329 In 857 InAs 867 InBr 862  $\ln Br_2$ 861  $InBr_3$ 859 InCl 862 InCl<sub>2</sub> 861 InCl<sub>3</sub> 858 InF<sub>3</sub> 228 InI 862 InI, 861 InI<sub>3</sub> 860 InN 866 In<sub>2</sub>O 863 In<sub>2</sub>O<sub>3</sub> 863 In(OH)<sub>3</sub> 862 InP 867 InS 864

 $\begin{array}{cccc} In_2S & 864 \\ In_2S_3 & 864 \\ InSb & 867 \\ InSe & 865 \\ In_2Se & 865 \\ In_2Se_3 & 865 \\ In_2Te & 865 \\ In_2Te_3 & 865 \\ In_2Te_3 & 865 \\ InF_4 & 271 \\ IrF_6 & 270 \\ \end{array}$ 

#### K

K 958 K<sub>3</sub>As 986 KAsH<sub>2</sub> 595 KBF4 223 KBF<sub>3</sub>OH 223K<sub>3</sub>Bi 986  $KBiO_3 \cdot \frac{1}{3}H_2O$ 628 KBrF<sub>4</sub> 237 KBrO · 3H<sub>2</sub>O 311KC<sub>8</sub> 635 KC<sub>24</sub> 635 KC<sub>36</sub> 635 KC<sub>48</sub> 635 K<sub>2</sub>CO<sub>3</sub> 987 K<sub>3</sub>COF<sub>7</sub> 269 269 $K_2 Cr F_6$  $K_3CUF_6$  269 KF 236 KF • HF 237 $K_2 FeF_6 = 269$ KGe 989 K<sub>2</sub>GeF<sub>6</sub> 216 KH 971 KHF<sub>2</sub> 146 KHPO<sub>3</sub>NH 579 KI 290 KIBr<sub>2</sub> 296 KICl<sub>2</sub> 295 KICl<sub>4</sub> 298 KIF<sub>6</sub> 238 KI<sub>3</sub> · H<sub>2</sub>O 294 KIO<sub>4</sub> 325  $K_2MnF_6$  264, 269 KN<sub>3</sub> 476

 $K(NH_3)_2C_{12}$  636  $\begin{array}{rrr} K_2 NbF_7 & 255 \\ K_2 NiF_6 & 269 \end{array}$ KO<sub>2</sub> 981 K<sub>2</sub>O 974 KPF<sub>6</sub> 196  $K_3PO_4 \cdot 8H_2O$ 545  $K_4 P_2 O_8 = 562$ K<sub>2</sub>PbCl<sub>6</sub> 753 KPbI<sub>3</sub> 754 KPbI₃ • 2H,O 754 K<sub>2</sub>S 360 K<sub>2</sub>S<sub>2</sub> 363  $K_2S_3$ 364 366  $K_2S_4$  $K_2S_5$  367  $K_{2}S_{6}$  368 KSCN 739  $K_2S_2O_8$ 392  $K_2S_3O_6$ 398 K<sub>2</sub>S<sub>4</sub>O<sub>6</sub> 399  $K_2S_5O_6 \cdot 1.5H_2O$  401  $K_2S_6O_6$  403  $KSO_2F$  178  $K_2SO_3(NO)_2$  504 K<sub>3</sub>Sb 986  $KSbCl_6 \cdot H_2O$  612 K<sub>2</sub>Se 421  $K_2 SeCl_6$  425 KSi 989  $K_2 SnCl_6$  731 K<sub>2</sub>TaF<sub>7</sub> 256 K<sub>2</sub>Te 441  $K_2 TeCl_6$  444  $K_2(TlCl_5H_2O) \cdot H_2O$ 874  $K_3(TlCl_6) \cdot 2H_2O$  873  $K_2 V F_6 = 269$ 

## L

LaF<sub>3</sub> 246 Li 956 Li<sub>3</sub>Al 830 Li<sub>3</sub>AlAs<sub>2</sub> 831 LiAl(CN)<sub>4</sub> 833 LiAlH<sub>4</sub> 680, 805  $Li_3AlN_2$  828  $Li_{3}AlP_{2}$  830  $Li_{3}As$  985  $LiBH_4$  775  $LiBH_{4} \cdot O(C_{2}H_{5})_{2}$  775 Li<sub>3</sub>Bi 985  $Li_2C_2$  987 Li<sub>2</sub>CO<sub>3</sub> 950, 987 LiF 235 LiGaH<sub>4</sub> 842 LiH 971, 805 LiN<sub>3</sub> 475 Li<sub>3</sub>N 984 LiNH<sub>2</sub> 463 Li<sub>2</sub>NH 464  $Li(NH_3)_2C_{12}$ 636 Li<sub>2</sub>O 974 Li<sub>2</sub>O<sub>2</sub> 975, 979 LiOH 983 LiOH • H<sub>2</sub>O 983 Li<sub>3</sub>P 985 Li<sub>3</sub>Sb 985 Li<sub>2</sub>Si 991 Li<sub>4</sub>Si 991 Li<sub>2</sub>SiO<sub>3</sub> 705

#### M

Mg(OH)<sub>2</sub> 912 917  $Mg_3P_2$ 913 MgS  $Mg_3Sb_2$ 606 915MgSe Mg<sub>2</sub>Si 921MgTe 915  $MnF_{2}$ 262 $MnF_3$ 263 259MOF<sub>6</sub> Ν N<sub>2</sub> 457  $N_3Br$ 477 $NBr_3 \cdot 6NH_3$ 480 NCl<sub>3</sub> 479  $N_3Cl 476$  $(NCl)_3(SO)_3$ 412 $NCl(SO_3K)_2$ 508 ND<sub>3</sub> 137  $NF_3$ 181 $NH_3$ 460<sup>15</sup>NH<sub>3</sub> 461  $N_2H_4$ 469  $N_2H_4 \cdot H_2O$  469  $NH_4AlF_4$  227  $(NH_4)_3AlF_6$ 226 $(NH_4)_3AsO_4 \cdot 3H_2O$ 602  $(NH_4)_3AsS_4$ 604  $(NH_4)_2BeF_4$ 232480  $NH_2Br$  $(NH_4)_2 CS_3 = 674$ NH<sub>2</sub>Cl 477 NH<sub>4</sub>ClO<sub>3</sub> 313  $NH_4F$  183  $NH_4F \cdot HF$ 183 $(NH_4)_3GaF_6 = 228$  $NH_4Ga(SO_4)_2 \cdot 12H_2O$ 854  $NH_4HPO_3NH_2$  584, 588  $(NH_4)_2HPO_3S$ 584NH₄HS 357  $(NH_4)_3InF_6$ 229 $NH_4I$  289

 $NH_2NO_2$  496 NH<sub>2</sub>OH 501  $(NH_3OH)_3AsO_4$ 501  $(NH_3OH)_2C_2O_4$ 501 (NH<sub>3</sub>OH)Cl 498 (NH<sub>3</sub>OH)HSO<sub>4</sub> 499  $(NH_3OH)_3PO_4$  500, 501NH<sub>2</sub>OSO<sub>3</sub>H 511 $NH_4PF_6$  195  $(NH_4PO_3)_X$ 580  $NH_4PO_2F_2$ 196  $(NH_{4})_{2}P_{2}O_{5}(NH_{2})_{2}$ 588  $(NH_4)_2PbCl_6$  751  $(NH_4)_2S_5$ 369  $NH_3SO_4$ 510 $(NH_4)_2S_2O_8$ 390  $N_2H_6SO_4$ 468 NH<sub>2</sub>SO<sub>3</sub>H 508  $NH(SO_3K)_2$ 506 NH<sub>2</sub>SO<sub>3</sub>K 507  $N_{2}H_{2}(SO_{3}K)_{2}$ 504, 509  $N_2H_2(SO_3NH_4)_2$ 509  $N_2H_2(SO_3Py)_2$ 510  $(NH_4)_2SbBr_6$ 615  $(NH_4)_2 SeCl_6$ 425 $(NH_4)_2 SnCl_6$ 731 $(NH_4)_2 TeCl_6$ 444481  $NI_3 \cdot NH_3$ NO 485  $NO_2$  488 N<sub>2</sub>O 484  $N_2O_3$ 487 $N_2O_4$ 488  $N_2O_5$ 489  $NOBF_4$ 2**2**4  $NOBF_4 \cdot H_2O$ 224 $NO_{2}[BF_{4}]$  187  $N_2O_5 \cdot BF_3$  187 NOBr 513 NOCL 511 NO<sub>2</sub>Cl 513 NOClO<sub>4</sub> 320  $NO_2ClO_4$ 321NOF 184

 $NO_2F$ 186187  $NO_3F$ (NO) HSO<sub>4</sub> 406  $NOH(SO_3K)_2$ 503  $NO_2NH_2$  497 NO<sub>2</sub>NHCOOK 497 NOSO<sub>2</sub>F 186 $NO(SO_3K)_2$ 504 NO(SbCl<sub>c</sub>) 612  $NO_{2}(SbCl_{6})$ 612 $N_2(SO_3K)_2$ 510 $N(SO_3K)_3 \cdot 2H_2O = 506$ Na 958 NaAlCl<sub>4</sub> 816 Na<sub>3</sub>As 986 NaAsH, 595  $Na_3AsO_2S_2 \cdot 11H_2O$ 605  $Na_3AsO_3S \cdot 12H_2O$ 605  $Na_3AsS_4 \cdot 8H_2O$ 604 $Na[B(C_6H_5)_4]$ 803 NaBF<sub>4</sub> 222  $NaBH_4$ 776  $NaBH_4 \cdot 2H_2O 777$ NaBO<sub>2</sub> 791, 793  $NaBO_2 \cdot 0.5H_2O$  791  $NaBO_2 \cdot 2H_2O$ 791 $NaBO_2 \cdot 4H_2O$ 791  $NaBO_2 \cdot H_2O_2 \cdot 3H_2O$ 796  $NaBO_3 \cdot 4H_2O$ 795  $NaB_5O_8 \cdot 5H_2O$ 795  $Na_2B_4O_7$ 794 $Na_2B_4O_7 \cdot 2H_2O$ 794  $Na_2B_4O_7 \cdot 4H_2O$ 794 793  $Na_2B_4O_7 \cdot 5H_2O$  $Na_2B_4O_7 \cdot 10H_2O 793$ 790  $Na_3BO_3$  $Na_4BeO_3$ 895 Na<sub>3</sub>Bi 986 NaBiO<sub>3</sub> 627  $NaBiO_3 \cdot nH_2O$ 628 $NaBrO \cdot 5H_{2}O$ 310 $Na_2C_2$ 987  $Na_2CO_3$  988 NaClO 309

 $NaClO \cdot 5H_2O$ 309  $NaClO_2 \cdot 3H_2O$ 312235NaF NaGe 989 NaGeH<sub>3</sub> 714 Na<sub>2</sub>GeH, 714 NaH 971 $NaH_2AsO_4 \cdot H_2O$  602 NaHB(OCH<sub>3</sub>)<sub>3</sub> 777 $Na_3H_2IO_6$ 323 $Na_{3}HP_{2}O_{7}$ 548 $NaH_2PO_4 \cdot 2H_2O 544$  $Na_2H_2P_2O_6 \cdot 6H_2O$ 560 NaHS 357 419 NaHSe  $Na_2H_4TeO_6$ 453 $Na_2H_4TeO_6 \cdot 3H_2O$ 454323 $NaIO_3$ 323, 324 NaIO<sub>4</sub>  $NaIO_4 \cdot 3H_2O$ 324NaN<sub>3</sub> 474 Na<sup>15</sup>N<sub>3</sub> 466NaNH<sub>2</sub> 465Na<sup>15</sup>NH<sub>2</sub> 466 $Na(NH_3)_2C_{12}$ 636 NaNO 514  $Na_2NO_2$  515  $Na_2N_2O_3$ 517  $Na_2N_2O_2 \cdot 9H_2O$ 495  $NaO_2$ 980  $Na_2O$ 974 979  $Na_2O_2$ 121NaOD Na<sub>3</sub>P 986  $NaPH_2$ 530  $NaPH_2 \cdot 2NH_3$ 530 Na<sub>2</sub>PH 530  $(NaPO_3)_X$ 549 (NaPO<sub>3</sub>)y 550  $(NaPO_3)_Z$ 551 $Na_{3}P_{3}O_{9} \cdot 6H_{2}O$ 552 $Na_4P_2O_6 \cdot 10H_2O 561$  $Na_4P_4O_{12} \cdot nH_2O$  553  $Na_5P_3O_{10}$ 547 $Na_5P_3O_{10} \cdot 6H_2O 547$ 

 $Na_6P_4O_{13}$  548  $Na_6P_4O_{13} \cdot aq$ 548 $(Na_2PO_3)_3N$ 590  $Na_2PO_3NH_2$ 588  $Na_2PO_3NH_2 \cdot 6H_2O$ 581 $Na_4P_2O_6NH$ 589  $Na_4P_2O_6NH \cdot 10H_2O$ 589  $Na_3POS_3 \cdot 11H_2O$ 571 $Na_3PO_2S_2 \cdot 11H_2O$ 570  $Na_3PO_3S \cdot 12H_2O$ 569  $NaPOS(NH_2)_2$ 589  $Na_3PS_4 \cdot 8H_2O$ 572 $Na_2PbO_3$ 758  $Na_4PbO_4$ 759  $Na_{2}S$ 358  $Na_2S_2$  361 365  $Na_2S_4$ 367  $Na_2S_5$  $Na_2S_2O_4 \cdot 2H_2O$ 393  $Na_2S_2O_6 \cdot 2H_2O$ 395 Na<sub>3</sub>Sb 986 619  $Na_3SbS_4 \cdot 9H_2O$  $Na_2Se$  421  $Na_2Se_2$  421 f  $Na_2SeO_3 \cdot 5H_2O$ 431  $Na_2SeO_4$  433 f  $Na_2SeS_4O_6 \cdot 3H_2O$ 434 NaSi 989  $Na_2SiO_3$ 704  $Na_2SiO_3 \cdot 9H_2O$ 704  $Na_2Si_2O_5$ 704742 $Na_2SnS_3 \cdot 8H_2O$  $Na_4SnO_4$ 739  $Na_4SnS_4 \cdot 18H_2O$  743  $Na_2Te$  441  $Na_2Te_2$  442  $Na_2TeO_3$ 449  $Na_2TeO_4$ 449, 453  $Na_6 TeO_6$ 453 $Na_6TeO_6 \cdot 2H_2O$  453  $Na_2TeS_4O_6 \cdot 2H_2O$  454  $\begin{array}{rrr} \mathrm{Nb}\,\mathrm{F}_5 & 254\\ \mathrm{Ne} & 82\\ \mathrm{Ni}\,\mathrm{F}_2 & 269 \end{array}$ 

#### Ð

 $\begin{array}{cccc} O_2 & 334 \\ O_3 & 337 \\ OF_2 & 163 \\ O_2F_2 & 162 \\ [OS(N)C1]_3 & 412 \end{array}$ 

#### P

P 518 PCl<sub>2</sub>F 191 PCl<sub>2</sub>F<sub>3</sub> 192  $PCl_4 \cdot PF_6$ 193 179, 189  $\mathbf{PF}_{3}$  $PF_5$ 190, 194  $PH_3$ 525 $P_2H_4$ 525PH₄I 531 $PI_3$ 540  $P_2I_4$ 539 574 $P_3N_5$ (PNBr<sub>2</sub>)<sub>n</sub> 578 (PNCl<sub>2</sub>)n 575  $(PNF_2)_3$ 194 $(PNF_2)_4$ 194P<sub>2</sub>O<sub>5</sub> 541, 825  $POBr_3$ 534 $P_2O_3Cl_4$ 536  $P_4O_4Cl_{10}$ 536 579  $POCl(OC_6H_5)_2$ POCl<sub>2</sub>(OC<sub>6</sub>H<sub>5</sub>) 582 POF<sub>3</sub> 179, 193  $PO(NH_{2})_{3}$ 584 $P_2O_3(NH_2)_4$ 588  $[PO(NH)NH_2]n$ 588  $[PO(NH_2)_2]NH$ 587 [PO(NH<sub>2</sub>)<sub>2</sub>NH]<sub>2</sub>PONH<sub>2</sub> 587  $PONH_2(OC_6H_5)_2$ 577 PO(NH<sub>2</sub>)<sub>2</sub>(OC<sub>6</sub>H<sub>5</sub>) 582 $P_2S_5$  567

 $P_{2}S_{5} \cdot 7NH_{3} = 574$ 563  $P_4S_3$  $P_4S_5$ 565  $P_4S_7$  566  $P_4S_{10}$ 567  $PSBr_3 535$  $PSBr_3 \cdot H_2O = 536$  $PSCl_3 532$  $PS(NH_{2})_{3}$  587  $P_4Se_3$  573 Pb 748  $Pb(CH_3)_4$  763  $Pb(C_2H_5)_4$  765 Pb(CH<sub>3</sub>COO)<sub>4</sub> 767 PbCO<sub>3</sub> 766  $2PbCO_3 \cdot Pb(OH)_2$ 767 PbCl<sub>4</sub> 750 218  $PbF_2$ PbF₄ 219 Pb(N<sub>3</sub>)<sub>2</sub> 763 757 PbO<sub>2</sub>  $Pb_{3}O_{4}$  755  $Pb_2P_2O_6$ 558 760 PbS Pb(SCN)<sub>2</sub> 769  $Pb(SO_4)_2$ 761705  $PbSiO_3$ R Rb 958 RbC<sub>8</sub> 635  $RbC_{24}$ 635 RbC<sub>36</sub> 635  $RbC_{48}$ 635  $Rb_2CO_3$  987 **RbCl** 951 RbGe 989 RbH 971 RbN<sub>3</sub> 476  $Rb(NH_3)_2C_1_2$ 637 **RbO**<sub>2</sub> 981 Rb<sub>2</sub>O 974 **RbOH** 983  $Rb_{2}S_{2}$  369  $Rb_2S_3$ 369

 $Rb_2S_5$  369  $Rb_2SeCl_6$  425 RbŠi 989  $Rb_{2}TeCl_{6}$  444  $Rb(TlBr_{4}) \cdot H_{2}O = 876$  $Rb_3(TlBr_6) \cdot \frac{8}{7}H_2O$ 876  $ReF_{e}$  264 S S 341  $S_2Br_2$ 377  $S_3Br_2$ 379  $S_4Br_2$ 379  $S_5Br_2$ 379  $S_6Br_2$ 379  $S_7Br_2$ 379  $S_8Br_2$ 379  $(SCN)_2$  671  $SCl_2$  370  $SCl_4$ 376S<sub>2</sub>Cl<sub>2</sub> 371  $S_3Cl_2$  373  $S_4Cl_2$  372, 375  $S_5Cl_2$  372, 375 S<sub>6</sub>Cl<sub>2</sub> 372, 375 S<sub>7</sub>Cl<sub>2</sub> 372, 376  $S_8Cl_2$ 372, 376 SF₄ 168 $SF_6$ 169  $S_2N_2$  409  $S_4N_2$ 408 406  $S_4N_4$  $S_4(NH)_4$  411 S<sub>7</sub>NH 411  $S_3N_2O_2$  413  $S_3N_2O_5$  414 SO 379  $(SO_{3-4})_{X}$ 382S<sub>2</sub>O 380  $S_2O_3 = 380$ SOBr<sub>2</sub> 387 SO<sub>2</sub>BrF 176 SOCl<sub>2</sub> 382 SO<sub>2</sub>Cl<sub>2</sub> 383  $S_2O_5Cl_2$  386

SOCIF 174SO<sub>2</sub>ClF 175 SOF<sub>2</sub> 170, 179 SOF<sub>4</sub> 171  $SO_2F_2$ 173 $S_3O_8F_2$  174  $(SO_2N)_3Ag_3 \cdot 3H_2O$ 483 SONH 480  $(SO_2NH)_3$  483 SO<sub>2</sub>(NH<sub>2</sub>), 482  $SO_{2}(NHAg)_{2}$ 483 Sb 606 SbBr<sub>3</sub> 613 SbCl<sub>3</sub> 608 SbCl 610  $SbCl_5 \cdot H_2O$  610  $SbCl_5 \cdot 4H_2O$  610  $SbCl_2F_3$  200 199  $SbF_3$  $SbF_5$ 143, 200  $SbH_3$ 606  $SbI_3 614$  $Sb_2O_3$ 615  $Sb_2O_4$ 618  $Sb_2O_5 616$  $Sb_2O_5 \cdot (H_2O)_X$ 617 SbOCl 611  $Sb_4O_5Cl_2$  611  $(SbO)_2SO_4$  619  $Sb_{2}(SO_{4})_{3}$ 618 ScF<sub>3</sub> 245 415Se  $SeBr_4$  427  $Se_2Br_2$  426  $SeCl_4$  423  $\begin{array}{ccc} \mathrm{Se}_2\mathrm{C}\bar{\mathrm{I}}_2 & 422\\ \mathrm{Se}\,\mathrm{F}_4 & 180^{\circ} \end{array}$ 179 SeF<sub>6</sub>  $Se_4N_4$  435 SeO<sub>2</sub> 428 SeOCl, 429  $SeO(OC_2H_5)_2$ 435SeSO<sub>3</sub> 435 Si 676 SiBr, 687 SiBr<sub>4</sub> 686, 688

 $Si_2Br_6$  688 Si(CH<sub>3</sub>COO)<sub>4</sub> 701SiCH<sub>3</sub>Cl<sub>3</sub> 695  $Si(CH_3)_2Cl_2$  694, 695 SiCl<sub>4</sub> 680, 682  $Si_2Cl_6$ 680  $Si_3Cl_8$  684 Si<sub>4</sub>Cl<sub>10</sub> 684  $Si_5Cl_{12}$  684  $\mathbf{684}$  $Si_6Cl_{14}$ Si<sub>10</sub>Cl<sub>22</sub> 684  $Si_{10}Cl_{20}H_2$  685  $SiF_4$  212  $(SiH)_X$  681  $(SiH_2)_X 681$  $SiH_4$  679, 680 679  $Si_2H_6$  $Si_{3}H_{8}$  679 SiHBr<sub>3</sub> 692  $SiH_2Br_2$  694 SiHCl<sub>3</sub> 691  $SiH_2Cl_2$  691 SiH<sub>3</sub>Cl 691 SiHF<sub>3</sub> 214 SiI<sub>4</sub> 689  $Si_2I_6$  690  $Si(N_3)_4$  476 Si(NCO)<sub>4</sub> 702 SiO 696 Si<sub>2</sub>O<sub>3</sub> 700 Si(OCH<sub>3</sub>)<sub>4</sub> 702  $Si(OC_2H_5)_4$ 702Si(OCN)<sub>4</sub> 702Si<sub>2</sub>OCl<sub>6</sub> 696  $Si_3O_2Cl_8$ 696  $Si_4O_3Cl_{10}$  696  $Si_4O_4Cl_8$ 695  $Si_5O_4Cl_{12}$  696  $Si_6O_5Cl_{14}$ 696  $Si_7O_6Cl_{16}$ 696 SiS 700  $SiS_2$  700 727 Sn  $SnBr_2$  732 733  $\operatorname{SnBr}_4$ Sn(CH<sub>3</sub>)<sub>4</sub> 744  $Sn(C_2H_5)_4$  746

 $Sn(CH_3COO)_4$ 747 SnCl<sub>2</sub> 728729 SnCl₄  $SnF_2$ 217 $SnF_4$ 217 $SnI_2$ 734 $SnI_4$ 735 SnO 736  $SnO_2$  738  $SnO_2 \cdot nH_2O$ 737 SnS 739 SnS<sub>2</sub> 741  $Sn(SO_4)_2 \cdot 2H_2O$ 744 Sr 922  $SrBr_2$  930  $SrCO_3$ 931930  $SrCl_2$  $Sr(ClO_4)_2$ 320 $Sr(ClO_4)_2 \cdot 4H_2O$  320  $\operatorname{Sr} \mathbf{F}_{2}$  234 SrH<sub>2</sub> 929  $SrI_2$  930  $Sr(N_3)_2$ 941  $\begin{array}{c} \mathbf{Sr}_{3}\mathbf{N}_{2} & 940\\ \mathbf{SrO} & 932 \end{array}$  $SrO_2$  936  $SrO_2 \cdot 8H_2O$  935  $Sr(OH)_2 \cdot 8H_2O$  935 SrS 938 SrSe 939  $SrSeO_3$ 939  $SrSeO_4$  939 SrSi 947 SrTe 940 Т 255 $TaF_5$ Te 437 TeBr₄ 445 TeCl<sub>4</sub> 442 TeF<sub>f</sub> 180 TeI4 447TeO, 447 TeO<sub>3</sub> 450  $Te_2O_3(OH)NO_3$ 437,

447

 $TeSO_3$  455  $\mathbf{248}$ TiF<sub>3</sub> 250 $TiF_4$  $TiH_2$ 114 Tl 867 TlBr 869  $TlBr_3 \cdot 4H_2O$ 874  $Tl_2CO_3$  884 TlCl 869 TlCl<sub>3</sub> 870  $TlCl_3 \cdot 4H_2O$ 870 TlF 230 TlF<sub>3</sub> 230 TlI 869  $TlI \cdot I_2 876$  $TlI_3$  876 Tl<sub>3</sub>N 883 TINO3 883 Tl<sub>2</sub>O 877  $Tl_2O_3$ 879  $Tl_2O_3 \cdot xH_2O$ 879 TlOH 877 Tl(OH)SO4 ·2H2O 882 Tl<sub>2</sub>(OOC)<sub>2</sub>CH<sub>2</sub> 884 TlOOCH 884Tl<sub>2</sub>S 880  $Tl_2S \cdot Tl_2S_3$ 880  $Tl_2SO_4$  881  $Tl_2Se \cdot Tl_2Se_3$ 881  $Tl_2TeCl_6$  444 Tl(TlBr<sub>4</sub>) 875 Tl<sub>3</sub>(TlBr<sub>f</sub>) 875 Tl(TlCl<sub>4</sub>) 872  $Tl_3(TlCl_e) = 873$ 

#### U

UD <sub>3</sub> UF4 UF6 UH3	123 261 262 113 <b>V</b>
VF <sub>3</sub>	252
VF <sub>4</sub>	252
VF <sub>5</sub>	253

#### W

WF<sub>6</sub> 260

# Y

YF<sub>3</sub> 246

# Z

 $\begin{array}{rrrr} {\rm Zn_3As_2} & 594 \\ {\rm Zn\,F_2} & 242 \\ {\rm ZnS_2O_4} & 394 \\ {\rm Zr\,F_4} & 251 \end{array}$ 

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# HANDBOOK OF Preparative Inorganic Chemistry

# VOLUME 2 · SECOND EDITION

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# Translation Editor's Preface

The English version of Volume II of Brauer's "Handbook" follows the path of the very well received translation of Volume I. Again, some of the material and particularly the bibliography has been corrected and brought up to date. The nomenclature has been revised where necessary, with the Stock and the Stock-Werner systems (the practice of using Roman numerals to define oxidation states of atoms) adopted as much as possible. This conforms with current I. U. P. A. C. and Chemical Abstracts practice [for details of this, see Robert C. Brasted, J. Chem. Education <u>35</u>, 136 (1948)]. The references to laboratory equipment and techniques reflect current U.S. usage, but useful European methods have been retained.

It is hoped that this volume will be as well received as the preceding one. Comments from users are invited to help improve future editions.

Paul G. Stecher

Rahway, N. J. May 1965

# Contents

TRANSLATION EDITOR'S PREFACE .... V

# Part II (continued)

# **Elements and Compounds**

SECTION 19. COPPER, SILVER, GOLD	1003
Copper (Pure Metal)	1003
Colloidal Copper	1003
Copper Hydride CuH	1004
Copper (I) Chloride CuCl	1005
Copper (I) Bromide CuBr	1006
Copper (I) Iodide CuI	1007
Copper (II) Chloride CuCl <sub>2</sub>	1008
Copper (II) Bromide CuBr <sub>2</sub>	1009
Copper Oxychloride $CuCl_2 \cdot Cu(OH)_2 \dots \dots$	1010
Copper (I) Oxide $Cu_2O$	1011
Copper (II) Oxide CuO	1012
Copper (II) Hydroxide Cu(OH) 2	1013
Potassium Cuprate (III) KCuO <sub>2</sub>	1014
Schweizer's Reagent	1016
Copper (I) Sulfide Cu <sub>2</sub> S	1016
Copper (II) Sulfide CuS	1017
Copper (I) Selenide Cu <sub>2</sub> Se	1019
Copper (I) Telluride Cu <sub>2</sub> Te	1019
Copper (I) Sulfate $Cu_2SO_4$	1020
Tetraamminecopper (II) Sulfate [Cu(NH <sub>3</sub> ) <sub>4</sub> ]SO <sub>4</sub> • H <sub>2</sub> O	1021
Copper (I) Nitride Cu <sub>3</sub> N	1021
Copper (II) Azide $Cu(N_3)_2$	1022
Copper Phosphide Cu <sub>3</sub> P	1023
Copper Diphosphide CuP <sub>2</sub>	1024
Basic Copper Carbonates CuCO <sub>3</sub> · Cu(OH) <sub>2</sub> (Green	
Cupric Carbonate)	1024
2 CuCO <sub>3</sub> · Cu(OH) <sub>2</sub> (Blue Cupric Carbonate)	1025
Copper (I) Acetylide $Cu_2C_2 \cdot H_2O$	1026
Paris Green (Copper Acetoarsenite)	1027
Fehling's Solution	1027
Very Pure Silver	1028
Silver Powder	1029
Silver from Residues	1029
Silver Mirrors	1031
Colloidal Silver	1034
Silver Iodide AgI	1035
Silver Chlorate AgClO <sub>3</sub>	1037

Silver Peroxide Ag <sub>2</sub> O <sub>2</sub> 1038         Sodium Orthoargentite Na <sub>3</sub> AgO <sub>2</sub> 1039         Silver (I) Selenide Ag <sub>2</sub> Se       1041         Silver (I) Telluride Ag <sub>2</sub> Te       1042         Silver Sulfate Ag <sub>2</sub> SO <sub>4</sub> 1042         Silver Sulfate Ag <sub>2</sub> SO <sub>3</sub> 1043         Silver Sulfate Ag <sub>2</sub> SO <sub>3</sub> 1043         Silver Amide AgN <sub>3</sub> 1044         Silver Aride Ag <sub>3</sub> N       1045         Silver Aride Ag <sub>3</sub> N       1045         Silver Caynamide Ag <sub>2</sub> CO <sub>3</sub> 1048         Silver Carbonate Ag <sub>2</sub> CO <sub>3</sub> 1048         Silver Tatrate Ag <sub>2</sub> CO <sub>4</sub> 1049         o-Phenathrolinesilver (II) Persulfate [Ag phen <sub>2</sub> ]S <sub>2</sub> O <sub>8</sub> 1050         Tris-α, α'-dipyridylsilver (II) Perchlorate       [Ag(dipyr) <sub>3</sub> ](ClO <sub>4</sub> ) <sub>2</sub> 1050         Very Pure Gold       1052       1052       1052         Colloidal Gold       1053       Gold (III) Choride AuCl <sub>3</sub> 1056         Gold (III) Choride AuCl <sub>3</sub> 1056       1056       1057         Potassium Tetrachloroaurate (III) KAuCl <sub></sub>	Silver Oxide Ag <sub>2</sub> O	1037
Sodium Orthoargentite Na <sub>3</sub> AgO <sub>2</sub>	Silver Peroxide Ag <sub>2</sub> O <sub>2</sub>	1038
Silver (I) Sulfide Ag <sub>2</sub> S	Sodium Orthoargentite Na <sub>3</sub> AgO <sub>2</sub>	1039
Silver (I) Selenide Ag <sub>2</sub> Se	Silver (I) Sulfide Ag <sub>2</sub> S	1039
Silver (i) Telluride $Ag_2Te$	Silver (I) Selenide Ag-Se	1041
Silver Sulfate $Ag_2SO_4$	Silver (I) Telluride Ago Te	1042
Silver Sulfite $Ag_{2}SO_{3}$	Silver Sulfate Ag SO4	1042
Silver Amide AgNI2.       1043         Silver Azide AgN3.       1044         Silver Nitride Ag3N.       1045         Silver Nitride Ag2N2.       1047         Silver Cyanamide Ag2CO3       1048         Silver Carbonate Ag2CO3       1048         Silver Carbonate Ag2CO3       1048         Silver Tartrate Ag2C4H4O6       1049         o-Phenanthrolinesilver (II) Persulfate [Ag phen2]S2O8       1050         Tris- $\alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate       1052         [Ag(dipyr)3](ClO4)2.       1052         Colloidal Gold       1052         Colloidal Gold       1053         Gold from Residues       1054         Gold (I) Chloride AuCl       1055         Gold (II) Chloride AuCl3.       1059         Gold (III) Chloride Au(Ol3.       1059         Gold (III) Oxide Au2O3       1059         Gold (III) Hydroxide Au(OH)3       1061         Fotassium Aurate KAuO2.       3H2O       1061         Gold (II) Sulfide Au2S.       1062         Gold (II) Sulfide Au2S.       1062         Gold (II) Sulfide Au2S.       1062         Gold (II) Sulfide Au2S.       1063         Gold (II) Sulfide Au2S.       1064         Gold (II) Sulfide	Silver Sulfite Ag <sub>2</sub> SO <sub>2</sub>	1043
Silver Azide AgN3       1045         Silver Nitride Ag3N       1046         Silver Acetylide Ag2C2       1047         Silver Cyanamide Ag2CN2       1047         Silver Carbonate Ag2CO3       1048         Silver Tartrate Ag3C 4H4O8       1049         o-Phenanthrolinesilver (II) Persulfate [Ag phen2]S2O8       1050         Tris $\neg \alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate       1050         [Ag(dipyr)3](ClO4)2       1050         Very Pure Gold       1052         Colloidal Gold       1053         Gold from Residues       1054         Gold (II) Chloride AuC1       1055         Gold (III) Chloride AuC1       1055         Gold (III) Chloride AuC1       1055         Gold (III) Oxide Au2O3       1059         Gold (III) Hydroxide Au(OH3       1061         Gold (II) Sulfide Au3       1061         Gold (II) Sulfide Au3       1062         Gold (II) Sulfide Au2S       1063         Gold (I) Cyanide Au2C       1063         Gold (I) Acetylide Au2C2       1063 <td>Silver Amide AgNH<sub>2</sub></td> <td>1043</td>	Silver Amide AgNH <sub>2</sub>	1043
Silver Nitride $Ag_3N$	Silver Azide AgNa	1045
Silver Acetylide $Ag_2C_2$	Silver Nitride AgaN	1046
Silver Cyanamide $Ag_2CN_2$ . 1047 Silver Carbonate $Ag_2CO_3$ . 1048 Silver Nitrite $Ag_2O_3$ . 1048 Silver Tartrate $Ag_2C_4H_4O_8$ . 1049 o-Phenanthrolinesilver (II) Persulfate [Ag phen <sub>2</sub> ]S <sub>2</sub> O <sub>8</sub> 1050 Tris- $\alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate [Ag(dipyr) <sub>3</sub> ](ClO <sub>4</sub> ) <sub>2</sub> . 1050 Very Pure Gold . 1052 Colloidal Gold . 1053 Gold from Residues . 1054 Gold (I) Chloride AuCl . 1055 Gold (III) Chloride AuCl <sub>3</sub> . 1056 Hydrogen Tetrachloroaurate (III) HAuCl <sub>4</sub> · 4 H <sub>2</sub> O . 1057 Potassium Tetrachloroaurate (III) KAuCl <sub>4</sub> · <sup>1</sup> / <sub>2</sub> H <sub>2</sub> O . 1058 Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub>	Silver Acetvlide Ag <sub>2</sub> C <sub>2</sub>	1047
Silver Carbonate $Ag_2CO_3$	Silver Cvanamide Ag <sub>2</sub> CN <sub>2</sub>	1047
Silver Nitrite AgNO <sub>2</sub>	Silver Carbonate Ag <sub>2</sub> CO <sub>2</sub>	1048
Silver Tartrate $Ag_2C_4H_4O_6$	Silver Nitrite AgNO <sub>2</sub>	1048
o-Phenanthrolinesilver (II) Persulfate [Ag phen <sub>2</sub> ]S <sub>2</sub> O <sub>8</sub> 1050 Tris- $\alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate [Ag(dipyr) <sub>3</sub> ](ClO <sub>4</sub> ) <sub>2</sub>	Silver Tartrate $Ag_{P}C_{A}H_{A}O_{A}$	1049
Tris $-\alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate       [Ag(dipyr)_3](ClO <sub>4</sub> ) <sub>2</sub> .       1050         Very Pure Gold       1052       Colloidal Gold       1052         Colloidal Gold       1053       Gold from Residues       1054         Gold (I) Chloride AuCl       1055       Gold (II) Chloride AuCl <sub>3</sub> .       1056         Hydrogen Tetrachloroaurate (III) HAuCl <sub>4</sub> · 4 H <sub>2</sub> O       1057       1058         Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub> 1059       1059         Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub> 1059       1060         Potassium Tetrachloroaurate (III) KAuCl <sub>4</sub> · <sup>1</sup> / <sub>2</sub> H <sub>2</sub> O       1058         Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub> 1069         Gold (III) Sulfide Au <sub>2</sub> O <sub>3</sub> 1061         Gold (II) Sulfide Au <sub>2</sub> S       1061         Gold (II) Sulfide Au <sub>2</sub> S       1063         Gold (I) Sulfide Au <sub>2</sub> S       1063         Gold (I) Sulfide Au <sub>2</sub> C <sub>2</sub> 1063         Gold (I) Cyanide AuCN       1064         Potassium Dicyanoaurate (I) K[Au(CN) <sub>2</sub> ]       1065         Section 20. Zinc, CADMIUM, MERCURY       1067         Zinc Zn       1067         Zinc Hydride ZnH <sub>2</sub> 1067         Zinc Hydroxychloride Zn(OH)Cl       1071         Ammonium Tetrachlorozincate (NH <sub>4</sub> ) <sub>2</sub> ZnCl <sub>4</sub> 1072      <	o-Phenanthrolinesilver (II) Persulfate [Ag phenalSaOa	1050
$\begin{bmatrix} Ag(dipyr)_3 \\ ClO_4 \\ 2 \\ 2 \\ 2 \\ Colloidal Gold$	Tris- $\alpha$ , $\alpha'$ -dipyridylsilver (II) Perchlorate	
Very Purg Gold       1052         Colloidal Gold       1053         Gold from Residues       1054         Gold (I) Chloride AuCl       1055         Gold (II) Chloride AuCla       1055         Gold (III) Chloride AuCla       1055         Gold (III) Chloride AuCla       1055         Gold (III) Chloride AuCla       1057         Potassium Tetrachloroaurate (III) HAuCl4 · 4 H <sub>2</sub> O       1057         Potassium Tetrachloroaurate (III) KAuCl4 · 1/2 H <sub>2</sub> O       1059         Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub> 1060         Potassium Aurate KAuO <sub>2</sub> · 3 H <sub>2</sub> O       1061         Gold (I) Sulfide Au <sub>2</sub> S       1061         Gold (II) Sulfide Au <sub>2</sub> S       1062         Gold (II) Sulfide Au <sub>2</sub> S       1063         Gold (I) Sulfide Au <sub>2</sub> S       1063         Gold (I) Cyanide AuCN       1064         Potassium Dicyanoaurate (I) K[Au(CN) <sub>2</sub> ]       1065         Section 20. Zinc, CADMIUM, MERCURY       1067         Zinc Hydride ZnH <sub>2</sub> 1069         Zinc Chloride ZnCl <sub>2</sub> 1070         Zinc Hydroxychloride Zn(OH)Cl       1071         Ammonium Tetrachlorozincate (NH <sub>4</sub> ) <sub>2</sub> ZnCl <sub>4</sub> 1072         Zinc Bromide ZnBr <sub>2</sub> 1073         Zinc Hydroxychloride Zn(C <sub>2</sub> <t< td=""><td><math display="block">[Ag(dipyr)_2](ClO_4)_2 + c + c + c + c + c + c + c + c + c + </math></td><td>1050</td></t<>	$[Ag(dipyr)_2](ClO_4)_2 + c + c + c + c + c + c + c + c + c + $	1050
Colloidal Gold	Very Pure Gold	1052
Gold from Residues	Colloidal Gold	1053
Gold (I) Chloride AuCl	Gold from Residues	1054
Gold (III) Chloride AuCl <sub>3</sub>	Gold (I) Chloride AuCl	1055
Hydrogen Tetrachloroaurate (III) HAuCl <sub>4</sub> · 4 H <sub>2</sub> O 1057 Potassium Tetrachloroaurate (III) KAuCl <sub>4</sub> · $^{1}/_{2}$ H <sub>2</sub> O 1058 Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub>	Gold (III) Chloride AuCla	1056
Potassium Tetrachloroaurate (III) KAuCl <sub>4</sub> · $1/2$ H <sub>2</sub> O       1058         Gold (III) Oxide Au <sub>2</sub> O <sub>3</sub> 1059         Gold (III) Hydroxide Au(OH) <sub>3</sub> 1060         Potassium Aurate KAuO <sub>2</sub> · 3 H <sub>2</sub> O       1061         Gold (I) Sulfide Au <sub>2</sub> S       1061         Gold (II) Sulfide Au <sub>2</sub> S       1062         Gold (II) Sulfide Au <sub>2</sub> S       1063         Gold (II) Sulfide Au <sub>2</sub> S       1063         Gold (I) Acetylide Au <sub>2</sub> C <sub>2</sub> 1063         Gold (I) Cyanide AuCN       1064         Potassium Dicyanoaurate (I) K[Au(CN) <sub>2</sub> ]       1065         SECTION 20. ZINC, CADMIUM, MERCURY       1067         Zinc Zn       1067         Zinc Hydride ZnH <sub>2</sub> 1069         Zinc Chloride ZnCl <sub>2</sub> 1070         Zinc Hydroxychloride Zn(OH)Cl       1071         Ammonium Tetrachlorozincate (NH <sub>4</sub> ) <sub>2</sub> ZnCl <sub>4</sub> 1072         Zinc Bromide ZnBr <sub>2</sub> 1073         Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub> 1074         Zinc Sulfide ZnS       1075         Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> 1076	Hydrogen Tetrachloroaurate (III) HAuClass 4 Hao	1057
Gold (III) Oxide $Au_2O_3$	Potassium Tetrachloroaurate (III) $KAuCl_{4} \cdot \frac{1}{2} H_{0}O$	1058
Gold (III) Hydroxide Au $(OH)_3$ 1060 Potassium Aurate KAu $O_2 \cdot 3 H_2O$ 1061 Gold (I) Sulfide Au $_2S$ 1062 Gold (II) Sulfide Au $_2S_3$ 1063 Gold (II) Sulfide Au $_2S_3$ 1063 Gold (I) Acetylide Au $_2C_2$ 1063 Gold (I) Cyanide AuCN 1064 Potassium Dicyanoaurate (I) K[Au $(CN)_2$ ] 1065 SECTION 20. ZINC, CADMIUM, MERCURY 1067 Zinc Zn 1067 Zinc Hydride ZnH $_2$ 1069 Zinc Chloride ZnCl $_2$ 1070 Zinc Hydroxychloride Zn $(OH)Cl$ 1071 Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072 Zinc Bromide ZnBr $_2$ 1073 Zinc Hydroxide (crystalline) $\epsilon$ -Zn $(OH)_2$ 1074 Zinc Sulfide ZnS 1076	Gold (III) Oxide Au <sub>2</sub> O <sub>2</sub>	1059
Potassium Aurate $KAuO_2 \cdot 3 H_2O$ 1061 Gold (I) Sulfide $Au_2S$ 1062 Gold (II) Sulfide $Au_2S_3$ 1063 Gold (II) Sulfide $Au_2S_3$ 1063 Gold (I) Acetylide $Au_2C_2$ 1063 Gold (I) Cyanide $AuCN$ 1064 Potassium Dicyanoaurate (I) K[Au(CN) <sub>2</sub> ] 1065 SECTION 20. ZINC, CADMIUM, MERCURY 1067 Zinc Zn 1067 Zinc Hydride ZnH <sub>2</sub> 1069 Zinc Chloride ZnCl <sub>2</sub> 1070 Zinc Hydroxychloride Zn(OH)Cl 1071 Ammonium Tetrachlorozincate (NH <sub>4</sub> ) <sub>2</sub> ZnCl <sub>4</sub> 1072 Zinc Bromide ZnBr <sub>2</sub> 1073 Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub> 1074 Zinc Sulfide ZnS 1075 Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> 1076	Gold (III) Hydroxide Au(OH)	1060
Gold (I) Sulfide Au2S.1061Gold (II) Sulfide Au2S.1062Gold (II) Sulfide Au2S31063Gold (I) Acetylide Au2C21063Gold (I) Cyanide AuCN1064Potassium Dicyanoaurate (I) K[Au(CN)2]1065SECTION 20. ZINC, CADMIUM, MERCURY1067Zinc Zn1063Zinc Chloride ZnCl21069Zinc Chloride ZnCl21070Zinc Hydroxychloride Zn(OH)Cl1071Ammonium Tetrachlorozincate (NH4)2Cl41072Zinc Iodide ZnI21073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	Potassium Aurate KAu $O_2 \cdot 3H_2O_3$	1061
Gold (II) Sulfide Aus Gold (II) Sulfide Au <sub>2</sub> S <sub>3</sub>	Gold (I) Sulfide Au <sub>2</sub> S.	1061
Gold (III) Sulfide $Au_2S_3$	Gold (II) Sulfide AuS	1062
Gold (I) Acetylide Au <sub>2</sub> C <sub>2</sub>	Gold (III) Sulfide Au <sub>2</sub> S <sub>2</sub>	1063
Gold (I) Cyanide AuCN	Gold (I) Acetvlide Au <sub>2</sub> C <sub>2</sub>	1063
Potassium Dicyanoaurate (I) $K[Au(CN)_2]$ 1065 SECTION 20. ZINC, CADMIUM, MERCURY 1067 Zinc Zn 1067 Zinc Hydride ZnH <sub>2</sub> 1067 Zinc Chloride ZnCl <sub>2</sub> 1069 Zinc Chloride ZnCl <sub>2</sub> 1070 Zinc Hydroxychloride Zn(OH)Cl 1071 Ammonium Tetrachlorozincate (NH <sub>4</sub> ) <sub>2</sub> ZnCl <sub>4</sub> 1072 Zinc Bromide ZnBr <sub>2</sub> 1072 Zinc Iodide ZnI <sub>2</sub> 1073 Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub> 1074 Zinc Sulfide ZnS 1075 Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> 1076	Gold (I) Cvanide AuCN	1064
SECTION 20. ZINC, CADMIUM, MERCURY. 1067 Zinc Zn 1067 Zinc Hydride ZnH <sub>2</sub> . 1069 Zinc Chloride ZnCl <sub>2</sub> . 1070 Zinc Hydroxychloride Zn(OH)Cl 1071 Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ . 1072 Zinc Bromide ZnBr <sub>2</sub> 1072 Zinc Iodide ZnI <sub>2</sub> 1073 Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub> 1074 Zinc Sulfide ZnS 1075 Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> 1076	Potassium Dicvanoaurate (I) K[Au(CN) <sub>2</sub> ]	1065
SECTION 20. ZINC, CADMIUM, MERCURY.       1067         Zinc Zn       1067         Zinc Hydride ZnH2       1069         Zinc Chloride ZnCl2       1070         Zinc Hydroxychloride Zn(OH)Cl       1071         Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072         Zinc Bromide ZnBr2       1072         Zinc Iodide ZnI2       1073         Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)2       1074         Zinc Sulfide ZnS       1075         Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)2       1076		
Zinc Zn1067Zinc Hydride ZnH21069Zinc Chloride ZnCl21070Zinc Hydroxychloride Zn(OH)Cl1071Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072Zinc Bromide ZnBr21072Zinc Iodide ZnI21073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	SECTION 20. ZINC, CADMIUM, MERCURY	1067
Zinc Hydride $ZnH_2$ 1069Zinc Chloride $ZnCl_2$ 1070Zinc Hydroxychloride $Zn(OH)Cl$ 1071Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072Zinc Bromide $ZnBr_2$ 1072Zinc Iodide $ZnI_2$ 1073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	Zinc Zn	1067
Zinc Chloride ZnCl21070Zinc Hydroxychloride Zn(OH)Cl1071Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072Zinc Bromide ZnBr21072Zinc Iodide ZnI21073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	Zinc Hydride ZnH <sub>2</sub>	1069
Zinc Hydroxychloride Zn(OH)Cl1071Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$ 1072Zinc Bromide ZnBr21072Zinc Iodide ZnI21073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	Zinc Chloride ZnCl <sub>2</sub>	1070
Ammonium Tetrachlorozincate $(NH_4)_2ZnCl_4$	Zinc Hydroxychloride Zn(OH)Cl	1071
Zinc Bromide $ZnBr_2$ 1072Zinc Iodide $ZnI_2$ 1073Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH)21074Zinc Sulfide ZnS1075Zinc Formaldehydesulfoxylate Zn(SO2 · CH2OH)21076	Ammonium Tetrachlorozincate $(NH_4)_2 ZnCl_4 \dots \dots$	1072
Zinc Iodide $ZnI_2$	Zinc Bromide ZnBr <sub>2</sub>	1072
Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub> 1074 Zinc Sulfide ZnS 1075 Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> 1076	Zinc Iodide ZnI <sub>2</sub>	1073
Zinc Sulfide ZnS	Zinc Hydroxide (crystalline) $\epsilon$ -Zn(OH) <sub>2</sub>	1074
Zinc Formaldehydesulfoxylate $Zn(SO_2 \cdot CH_2OH)_2 \dots 1076$	Zinc Sulfide ZnS	1075
	Zinc Formaldehydesulfoxylate Zn(SO <sub>2</sub> · CH <sub>2</sub> OH) <sub>2</sub> · · · · ·	1076

Ammonium Zinc Sulfate $(NH_4)_2 Zn(SO_4)_2 \cdot 6H_2O$	• •	1077
Zinc Selenide ZnSe	• •	1078
Zinc Amide Zn(NH <sub>2</sub> ) 2		1079
Zinc Nitride ZnaNa		1080
Zinc Phosphides Zn <sub>2</sub> P <sub>2</sub> , ZnP <sub>2</sub>		1080
Zinc Phosphate $Zn_2(PO_4)_2 \cdot 4H_2O$		1081
Zinc Hydroxyphosphate Zn <sub>2</sub> (OH)PO <sub>4</sub>		1082
Zinc Arsenides Zn <sub>2</sub> As <sub>o</sub> , ZnAs <sub>o</sub>		1083
Zinc Thioantimonate $Zn_{\rm c}({\rm SbS}_4)$		1083
Diethylzinc Zn(CoH_)		1084
Zinc Carbonate ZnCO <sub>2</sub>		1086
Zinc Acetate Zn(CHoCOO)		1087
Zinc Cyanide Zn(CN)		1087
Potassium Tetracyanozincate K-Zn(CN)		1088
Zinc Silicate Zn-SiO		1089
Zine Fluerosilicate ZnSiF. 6H-O	•••	1090
Zine Fluorosificate Zinfrg $0.112001200120$	••	1090
$\operatorname{Pinmann's Green}$	••	1000
Cadmium (needles) Cd	••	1002
Cadmium (hlorida CdCl	••	1002
Cadmium Undrowychlanida Cd(OH)Cl	• •	1000
Detaggium Cadmium Chlorido CdCl KCl H O	••	1094
Potassium Caumium Chioride CuCi <sub>2</sub> · KCi · $\Pi_2$ O · · ·	••	1000
Cadmium Iodida Cdl	••	1000
Cadmium Hudnovido Cd(OH)	••	1000
Cadmium Sulfide CdS	••	1007
Cadmium Nitrido Cd N	• •	1100
Cadmium Amida $(d/NH)$	••	1100
Caumium Amide $Ca(NH_2)_2$	••	1100
Cadmium Phosphildes $Cd_3P_2$ , $CdP_2$ , $CdP_4$	••	1101
Distributes $G_3AS_2, GAS_2 \dots \dots \dots$	••	1100
Dietnyicaamium $Ca(C_2H_5)_2$	••	1103
Cadmium Carbonate CdCO3 $\dots \dots \dots \dots \dots \dots \dots$	••	1104
Cadmium Acetate $Cd(CH_3COO)_2$	••	1105
Cadmium Cyaniae $Ca(CN)_2$	• •	1100
Polassium Teiracyanocadmale $K_2Cd(CN)_4$	••	1100
Cadmium Thiocyanate Cd(SCN) <sub>2</sub>	••	1100
Cadmium Silicate $Cd_2SiO_4$	••	1107
Cadmium Ferrate (III) CaFe <sub>2</sub> O <sub>4</sub>	••	1107
Mercury (II) Oxychloride $HgCl_2 \cdot 4 HgO \dots$	••	1108
Mercury (II) Bromide HgBr <sub>2</sub>	••	1109
Potassium Triiodomercurate (II) $KHgl_3 \cdot H_2O$	••	1110
Copper (1) Tetraiodomercurate (II) $Cu_2HgI_4$	••	1110
Mercury (II) Sulfide HgS	••	1111
Mercury (II) Selenide HgSe	••	1113
Mercury (II) Amide Chloride HgNH <sub>2</sub> Cl.	••	1114
Diamminemercury (II) Dichloride $HgCl_2 \cdot 2NH_3 \ldots$	••	1114
Mercury (II) Iminobromide Hg <sub>2</sub> (NH)Br <sub>2</sub>	• •	1115

Millon's Base NHg <sub>2</sub> OH · xH <sub>2</sub> O	1116
Bromide of Millon's Base NHg <sub>2</sub> Br	1117
Mercury (I) Thionitrosylate $[Hg_2(NS)_2]_x$	1117
Mercury (II) Thionitrosylate [Hg(NS) <sub>2</sub> ] <sub>x</sub>	1118
Diethylmercury $Hg(C_2H_5)_2$	1118
Mercury (I) Acetate $Hg_2(CH_3COO)_2$	1120
Mercury (II) Acetate Hg(CH <sub>3</sub> COO) <sub>2</sub>	1120
Mercury (II) Cyanide Hg(CN) <sub>2</sub>	1121
Potassium Tetracyanomercurate (II) K <sub>2</sub> Hg(CN) <sub>4</sub>	1122
Mercury (I) Thiocyanate Hg <sub>2</sub> (SCN) <sub>2</sub>	1122
Mercury (II) Thiocyanate Hg(SCN) 2	1123
Potassium Tetrathiocyanomercurate (II) $K_2Hg(SCN)_4$ .	1124
SECTION 21. SCANDIUM, YTTRIUM, RARE EARTHS	1125
Pure Scandium Compounds	1125
Treatment of Monazite Sand	1127
Treatment of Gadolinite	1129
Pure La. Pr and Nd Compounds from Cerium Earths by	
Ion Exchange	1131
Pure Cerium Compounds	1132
Pure Samarium Compounds	1135
Pure Europium Compounds	1136
Pure Ytterbium Compounds	1138
Metallic Rare Earths	1141
Bare Earth Trichlorides LnCl. (anhydrous)	1146
Bare Earth Tribromides LnBr. (anhydrous)	1148
Bare Earth Trijodides Lule (anhydrous)	1149
Rare Earth Dibalides InX. (anhydrous)	1150
Carium (III) Oxide Ce.O.	1151
Drageodymium (IV) Oxide DrO	1151
$\begin{array}{c} \text{Pare Forth Hydroxides I n(OH)}  (arystalling) \end{array}$	1159
Lanthanum Sulfida La S	1153
Lanthanum Solonidos La So. La So.	1154
La Co Dr and Nd Monophalooromidos Ins. Inse. Inse.	1155
Europium (II) Chalaggerides Euro Euro	1155
Baro Forth Sulfator In (SO ) and O	1156
Rate Earth Suffaces $Ln_2(504)_3 \cdot In_20 \dots \dots \dots \dots$	1157
Rare Earth Nitrates $L_{\rm IN}$ (an hydrous)	1150
Rare Earth Nilrates $Ln(NO_3)_3$ (annyarous)	1150
Rare Earth Cyclopentadienides $Ln(C_B H_5)_3 \dots \dots \dots$	1109
SECTION 22. TITANIUM, ZIRCONIUM, HAFNIUM, THORIUM	1161
Titanium Ti	1161
Zirconium, Hafnium Zr, Hf	1172
Thorium Th	1174
Separation of Zirconium and Hafnium	1179
Titanium, Zirconium and Thorium Hydrides	1184
Titanium (II) Chloride, Bromide and Iodide TiCl <sub>2</sub> , TiBr <sub>2</sub> ,	
Til <sub>2</sub>	1185

Titanium (III) Chloride, Bromide and Iodide TiCl3.	
TiCl <sub>3</sub> • 6 H <sub>2</sub> O; TiBr <sub>3</sub> , TiBr <sub>3</sub> • 6 H <sub>2</sub> O; Til <sub>3</sub>	1187
Titanium (IV) Chloride TiCl	1195
Ammonium Hexachlorotitanate (NH <sub>4</sub> ) [TiCl <sub>2</sub> ]	1199
Titanium (IV) Bromide TiBr <sub>4</sub>	1201
Zirconium (IV), Hafnium (IV) and Thorium (IV) Chlorides	
and Bromides ZrCl <sub>4</sub> , HfCl <sub>4</sub> , ThCl <sub>4</sub> ; ZrBr <sub>4</sub> , HfBr <sub>4</sub> ,	
ThBr.	1203
Thorium Chloride ThCl <sub>4</sub> $\cdot$ 8 H <sub>2</sub> O.	1204
Titanium (IV), Zirconium (IV) and Thorium (IV) Iodides	
$TiI_4$ ( $ZrI_4$ , $ThI_4$ )	1205
Titanium (III) Oxychloride TiOC1	1209
Titanium (IV) Oxychloride TiOCl <sub>2</sub>	1209
Zirconium Oxychloride $ZrOCl_2 \cdot 8 H_2O$	1210
Hafnium Oxychloride	1213
Lower Titanium Oxides TiO, Ti <sub>2</sub> O <sub>3</sub>	1214
Titanium (IV) Oxide TiO <sub>2</sub>	1216
Titanium (IV) Oxide Hydrate TiO <sub>2</sub> • n H <sub>2</sub> O	1218
Peroxotitanic Acid H <sub>4</sub> TiO <sub>5</sub>	1219
Zirconium (IV) Oxide ZrO <sub>2</sub>	1220
Hafnium (IV) Oxide HfO <sub>2</sub>	1221
Thorium (IV) Oxide ThO <sub>2</sub>	1221
Titanium, Zirconium and Thorium Sulfides $TiS_3$ , $TiS_2$ ,	
$TiS_{<2}$	1222
Zirconium Sulfides	1226
Thorium Sulfides	1226
Titanium (III) Sulfate $Ti_2(SO_4)_3$	1226
Titanoxy Sulfate TiOSO <sub>4</sub>	1228
Zirconium Sulfates	1231
Purification of Zr Salts via the Tetrahydrate	1232
Titanium, Zirconium and Hafnium Nitrides TiN, ZrN, HfN	1233
Thorium Nitride $Th_3N_4$	1236
Titanium Tetranitrate $Ti(NO_3)_4$	1237
Thorium Nitrate $Th(NO_3)_4 \cdot n H_2O$	1238
Titanium Oxonitrate, Zirconium Oxonitrate $TiO(NO_3)_2$ ,	
$\operatorname{ZrO}(\operatorname{NO}_3)_2$	1241
Titanium Phosphide, Zirconium Phosphides, Thorium	1041
Phosphilde TiP, $ZrP_2$ , $ZrP$ , $Tn_3P_4$	1241
Zirconium and Hafnium Phosphates	1244
Intanium, Zirconium and Hamium Carbides TiC, ZrC,	1045
HIU	1245
Thorium Carbides Thu, Thu <sub>2</sub>	1248
Thanium, Zirconium and Thorium Shicides 1151 <sub>2</sub> , ZrSi <sub>2</sub> ,	1040
1101 <sup>5</sup> • • • • • • • • • • • • • • • • • • •	1249
SECTION 23 VANADUUM NUODUUM TANTALUM	1252
Vanadium V	1252
·	1000

Vanadium (II) Chloride VCl <sub>2</sub>	1255
Vanadium (III) Chloride VCl <sub>3</sub> , VCl <sub>3</sub> $\cdot$ 6 H <sub>2</sub> O	1256
Vanadium (IV) Chloride VCl <sub>4</sub>	1259
Vanadium (II) Bromide VBr <sub>2</sub>	1260
Vanadium (III) Bromide VBr <sub>3</sub>	1260
Vanadium (II) Iodide VI <sub>2</sub>	1261
Vanadium (III) Iodide VI3	1262
Vanadium Oxychloride VOC1	1262
Vanadium Oxydichloride VOCl <sub>2</sub>	1263
Vanadium Oxytrichloride VOCla	1264
Vanadium Dioxychloride VO <sub>2</sub> Cl.	1265
Lower Vanadium Oxides	1266
Vanadium (III) Hydroxide V(OH)	1268
Vanadium (V) Oxide V <sub>2</sub> O <sub>5</sub>	1270
Ammonium Metavanadate NH VO2	1272
Alkali Vanadates	1273
Vanadium Sulfides	1274
Vanadium Selenides	1276
Vanadium (II) Sulfate $VSQ_4 \cdot 6 H_2Q_4$	1277
Hydrogen Disulfatovanadate (III) $HV(SO_4)_{\circ} \cdot 4 H_{\circ}O_{\circ}$	1282
Ammonium and Potassium Disulfatovanadate (III)	
$NH \cdot V(SO_4) = KV(SO_4) =$	1283
Vanadium (IV) Oxysulfate (Vanadyl Sulfate) VOSO	1285
Vanadium Nitrides	1286
Vanadium Phosphides VP, VP VP.	1287
Vanadium Carbides VC V C	1288
Dibenzenevanadium $(0)$ V(C $_{\rm eH_{o}}$ )	1289
Potassium Hexathiocvanatovanadate (III) KoV(SCN)	1291
Niobium Metal Tantalum Metal	1292
Vanadium Niobium and Tantalum Hydrides	1295
Nichium (II) Chloride NhCl.	1296
Niobium (III) Chloride NbCl	1297
Nichium (IV) Chloride NbCl.	1299
Tantalum (IV) Chloride TaCl.	1301
Niobium (V) and Tantalum (V) Chlorides NbCl- TaCl-	1302
Nichium Oxytrichlaride NbOCL	1307
Niobium (III) Bromide NbBr.	1309
Tantalum (IV) Bromide TaBr.	1310
Nichium (V) and Tantalum (V) Bromides NhBr-	1010
To $B_{n-1}$	1211
Nichium Avutnihnomida NhORn	1212
Niobium (IV) Niobium (III) and Niobium (II) Indidag	1919
Mobium (IV), Mobium (III) and Mobium (II) Iodides	1914
Nichium (V) Iodido NhI-	1015
Tantalum (V) Iodide Tal-	1916
Nichium (II) Ovide NhO	1917
Nichium (IV) Orido NhO	1010
	1910

Niodium (V) and Tantalum (V) Oxides $Nb_2O_5$ , $Ta_2O_5$ . Alkali Niobates and Tantalates	$\begin{array}{c} 1318\\ 1323 \end{array}$
$HTaO_4 \cdot n H_2O$ Potassium Perovyniobate Potassium Perovytantalate	1324
K <sub>3</sub> NbO <sub>8</sub> , K <sub>3</sub> TaO <sub>8</sub>	1325
Niobium and Tantalum Sulfides	1327
Niobium and Tantalum Nitrides	1328
Niobium and Tantalum Phosphides NbP <sub>2</sub> , TaP <sub>2</sub> , NbP,	
TaP	1330
Niobium and Tantalum Carbides	1331
SECTION 24. CHROMIUM, MOLYBDENUM, TUNGSTEN, URANIUM.	1334
Chromium Cr	1334
Chromium (II) Chloride CrCl <sub>2</sub>	1336
Chromium (III) Chloride CrCl <sub>3</sub>	1338
Chromium (II) Bromide CrBr <sub>2</sub>	1340
Chromium (III) Bromide CrBr <sub>3</sub>	1341
Chromium (II) Iodide CrI <sub>2</sub>	1341
Chromium (III) Iodide CrI <sub>3</sub>	1344
Chromium (III) Hydroxide Cr(OH) <sub>3</sub> . n H <sub>2</sub> O	1345
Chromium Sulfides CrS, $Cr_2S_3$	1346
Chromium Nitride CrN	1347
Hexaaquochromium (III) Chloride [Cr(OH <sub>2</sub> ) <sub>6</sub> ]Cl <sub>3</sub> Chloropentaaquochromium (III) Chloride	1348
[CrCl(OH <sub>2</sub> ) <sub>5</sub> ]Cl <sub>2</sub> · H <sub>2</sub> O Hexaamminechromium (III) Chloride and Nitrate	1350
[Cr(NH <sub>3</sub> ) <sub>6</sub> ]Cl <sub>3</sub> , [Cr(NH <sub>3</sub> ) <sub>6</sub> ](NO <sub>3</sub> ) <sub>3</sub> Chloropentaamminechromium (III) Chloride	1351
$[CrCl(NH_3)_5]Cl_2$ Triethylenediaminechromium (III) Sulfate, Chloride and Thiocyanate $[Cren_3]_2(SO_4)_3$ , $[Cren_3]Cl_3 \cdot 3.5 H_2O$ ,	1352
$[Cr en_3](SCN)_3 \cdot H_2O$ cis-Dichlorodiethylenediaminechromium (III) Chloride	1354
$[CrCl_2 en_2]Cl \cdot H_2O.$ trans-Dithiocyanatodi (ethylenediamine)chromium (III)	1356
trans-Dichlorodi(ethylenediamine)chromium (III)	1357
Chloride [CrCl <sub>2</sub> en <sub>2</sub> ]Cl Dichloroaquotriamminechromium (III) Chloride	1357
$[CrCl_2(OH_2)(NH_3)_3]Cl$ Hexaureachromium (III) Chloride $[Cr(OCN_2H_4)_6]Cl_3$ .	1358
$3 H_2O$ Rhodochromium Chloride [(NH <sub>3</sub> ) <sub>5</sub> Cr(OH)Cr(NH <sub>3</sub> ) <sub>5</sub> ]Cl <sub>5</sub>	$1359 \\ 1359$
Erythrochromium Chloride $[(NH_3)_5Cr(OH)Cr(NH_3)_4(OH_2)]Cl_5$	1360

Tris (2.2'-dipyridyl)chromium (II) Perchlorate	
$[Cr(dinv)_0](C]O_1)_0$	1361
Tris(2,2'-dipyridyl)chromium (I) Perchlorate	
$[Cr(dipy)_{0}]ClO_{4}$	1362
$Tris(2,2 - dipyridyl)chromium (0) [Cr(dipy)_2]$	1363
Hexaphenvlisonitrilochromium (0) [Cr(C_H5NC)_c]	1363
Chromium Orthophosphate CrPO	1364
Chromium (II) Sulfate $CrSO_4 \cdot 5 H_2O_5$	1365
Chromium (II) Salt Solutions	1366
Chromium (II) Acetate $Cr_2(CH_3COO)_4 \cdot 2H_2O_4$	1368
Chromium (II) Oxalate $CrC_{2}O_{4} \cdot 2 H_{2}O_{4}$	1370
Hexaaquochromium (III) Acetate $[Cr(OH_2)_6](CH_3COO)_3$	1371
Dihydroxohexaacetatotrichromium (III) Acetate and	
Chloride $[Cr_3(OH)_2(CH_3COO)_6](CH_3COO) \cdot n H_2O_1$	
$[Cr_{3}(OH)_{2}(CH_{3}COO)_{6}]Cl \cdot 8 H_{2}O$	1371
Potassium Trioxalatochromate (III) $K_3[Cr(C_2O_4)_3]$ .	
3 H <sub>2</sub> O	1372
Potassium Hexacyanochromate (III) K <sub>3</sub> [Cr(CN) <sub>6</sub> ]	1373
Potassium Hexathiocyanatochromate (II)	
$K_3[Cr(SCN)_6] \cdot 4 H_2O$	1374
Trilithium Hexaphenylchromate (III) $Li_3Cr(C_6H_5)_6$ .	
2.5 (C <sub>2</sub> H <sub>5</sub> ) <sub>2</sub> O	1375
Ammonium Tetrathiocyanatodiamminechromate (III)	
$\operatorname{NH}_{4}[\operatorname{Cr}(\operatorname{SCN})_{4}(\operatorname{NH}_{3})_{2}] \cdot \operatorname{H}_{2}O$	1376
Tetrathiocyanatodiamminechromic (III) Acid	
$H[Cr(SCN)_4(NH_3)_2]$	1377
Ammonium Tetrathiocyanatodianilinochromate (III)	
$NH_{4}[Cr(SCN)_{4}(C_{6}H_{5}NH_{2})_{2}] \cdot 1^{1}/_{2} H_{2}O$	1378
Potassium Tetrathiocyanatodipyridinochromate (III)	
$K[Cr(SCN)_4py_2] \cdot 2 H_2O$	1379
Trichlorotriaquochromium $[CrCl_3(OH_2)_3]$	1380
Trichlorotriethanolochromium $[CrCl_3(C_2H_5OH)_3]$	1380
Trichlorotriamminechromium $[CrCl_3(NH_3)_3]$	1381
Trichlorotripyridinechromium [CrCl <sub>3</sub> py <sub>3</sub> ]	1381
Chromium (III) Glycinate $(H_2NCH_2COO)_3Cr$	1382
Chromium (III) Xanthate $[C_{2}H_{5}OCS_{2})_{3}Cr]$	1383
Chromium (III) Acetylacetonate $(C_5H_7O_2)_3Cr$	1383
Chromyl Chloride CrO <sub>2</sub> Cl <sub>2</sub>	1384
Chromium Trioxide-Pyridine $CrO_3 \cdot 2$ py	1385
Chromyl Nitrate $CrO_2(NO_3)_2$	1386
Chromyl Perchlorate $CrO_2(ClO_4)_2$	1387
Rubidium Chromate Rb <sub>2</sub> CrO <sub>4</sub>	1300
Rubicium Dichromate KD $_2$ Ur $_2$ U $_7$	1200
Consister Disknesses $C_2$ $C_3$ $C_4$ $C_4$ $C_5$ $C$	1209
Detaggium Eluorochromate $K[C=0, E]$	1300
Potassium Chlorochromate K[CrO_Cl]	1300
rotassium chiorochromate A[CrO3CI]	1030

Potassium Tetraperoxochromate (V) K <sub>3</sub> CrO <sub>8</sub>	1391
$(NH_{4})$ o Cr o Q to $\frac{2}{12}$ H o Q	1392
Diperoxotriamminechromium (IV) (NH <sub>o</sub> ) <sub>o</sub> CrO	1392
Barium Orthochromate (IV) Ba CrO.	1393
Barium Chromate $(V)$ Ba $(CrO)$	120/
Darium Chromate $(V)$ Dag $(CrO_4)_2$	1204
Solium Infochromite NaCrS <sub>2</sub>	1094
Dibenzenechromium (0) ( $C_{eff}_{6}$ ) 2 Cr	1395
Bis (dipnenyl) chromium (0) ( $C_{12}H_{10}$ ) gCr	1390
Dibenzenechromium (1) lodide $[(C_6H_6)_2Cr]]$ .	1397
Bis(diphenyl)chromium (l) lodide $[Cr(C_{12}H_{10})_2]1$	1397
(Diphenyl)(benzene)chromium (I) Iodide	
$[(C_{12}H_{10})Cr(C_{6}H_{6})]I$	1398
Molybdenum Mo	1401
Dibenzenemolybdenum (0) (C $_{6}H_{6}$ ) $_{2}Mo$	1402
Molybdenum (II) Chloride Mo <sub>3</sub> Cl <sub>6</sub>	1403
Molybdenum (III) Chloride MoCl <sub>3</sub>	1404
Molvbdenum (V) Chloride MoCl <sub>5</sub>	1405
Molybdenum (III) Bromide MoBro	1407
Tribromotripyridinemolybdenum [MoBropyo]	1408
Potassium Heyachloromolybdate (III) K. MoCl.	1408
Molybdenum (IV) Oxide MoO	1400
Molybdenum (IV) Oxide MoO2	1/10
$\gamma$ -Molybuenum Oxide Mo $_4$ O $_{11}$	1/11
Moleh damum (VII) Origina Mag	1411
Molybaenum (VI) Oxide $MoO_3$	1412
Molybdic Acid $H_2MOO_4 \cdot H_2O_4$	1412
Ammonium Oxopentachloromolybdate (V)	
$(NH_4)_2[MOOC1_5]$	1413
Potassium Hydrogen Diperoxomonomolybdate	
$\operatorname{KHMoO}_{6} \cdot 2 \operatorname{H}_{2}O$	1414
Tetraamminezinc Tetraperoxomolybdate (VI)	
$[Zn(NH_3)_4]MoO_8$	1414
Molybdenum (IV) Sulfide MoS <sub>2</sub>	1415
Ammonium Tetrathiomolybdate $(NH_4)_2MoS_4$	1416
Potassium Octacyanomolybdate (IV)	
$K_{A}[Mo(CN)_{B}] \cdot 2 H_{2}O$	1416
Tungsten W.	1417
Tungsten (V) Chloride WCl =	1419
Tungsten (VI) Chloride WCL	1420
Tungsten (IV) Oxide WO.	1421
$N_{\rm Tungsten}$ Ovide W O	1499
Tungsten Blue H. $WO_{-}$	1499
Tungston (VI) Ovido WO	1/00
Vollow Tungatia Aaid H WO	1/0/
Tentow Tungstic Acia $\pi_2 W \cup_4 \dots \dots \dots \dots \dots \dots \dots$	1444
Tungsten Oxytetrachioride $WOUI_4$	1425
Tungsten (IV) Sullide $WS_2$	1425
Tungsten Hexaphenoxide W(OC 6H 5) 6	1426

Potassium Enneachloroditungstate (III) K <sub>3</sub> W <sub>2</sub> Cl <sub>2</sub>	1427
Hexachlorotripyridineditungstate (III) W <sub>2</sub> Cl <sub>6</sub> py <sub>3</sub>	1429
Potassium Octacyanotungstate (IV) K <sub>4</sub> [W(CN) <sub>B</sub> ] • 2H <sub>2</sub> O	1429
Potassium Octacyanotungstate (V) K <sub>3</sub> [W(CN) <sub>B</sub> ] • H <sub>2</sub> O	1430
Uranium U	1431
Uranium Hydride UH <sub>3</sub>	1434
Uranium (III) Chloride UCl <sub>3</sub>	1435
Uranium (IV) Chloride UCl <sub>4</sub>	1436
Uranium (V) Chloride UCl <sub>5</sub>	1438
Uranyl Chloride UO <sub>2</sub> Cl <sub>2</sub>	1439
Uranium (IV) Bromide UBr <sub>4</sub>	1440
Uranium (IV) Oxide UO <sub>2</sub>	1442
Uranium (VI) Oxide $UO_3$	1442
Alkali Uranates (VI) Li <sub>2</sub> UO <sub>4</sub> , Na <sub>2</sub> UO <sub>4</sub> , K <sub>2</sub> UO <sub>4</sub>	1445
Alkali Uranates (V) LiUO <sub>3</sub> , NaUO <sub>3</sub>	1445
Uranium Peroxide $UO_4 \cdot 2 H_2O_4$	1446
Uranium (IV) Sulfide US2	1446
Uranium (IV) Sulfate $U(SO_4)_2 \cdot 8 H_2O$ or $4 H_2O$	1447
Ammonium Uranyl Carbonate $(NH_4)_4[UO_2(CO_3)_3]$	1449
Uranium (IV) Oxalate $U(C_2O_4)_2 \cdot 6 H_2O_4$	1449
Potassium Tetraoxalatouranate (IV)	
$K_4[U(C_2O_4)_4] \cdot 5 H_2O_4$	1450
Uranium (V) Ethoxide $U(OC_{2}H_{5})_{5}$	1451
Uranium (VI) Ethoxide $U(OC_{2}H_{5})_{6}$	1452
Uranyldibenzoylmethane $UO_2(C_{15}H_{11}O_2)_2$	1453
SECTION 25. MANGANESE	1454
Manganese Mn.	1454
Manganese (II) Oxide MnO	1455
Manganese (II) Hydroxide Mn(OH)	1456
Manganese (III) Oxide $\gamma$ -Mn <sub>2</sub> O <sub>3</sub> , $\gamma$ -MnO(OH)	1457
Manganese (IV) Oxide MnO <sub>2</sub>	1458
Manganese (VII) Oxide $Mn_2O_2$	1459
Sodium Manganate (V) Na $_{3}$ MnO <sub>4</sub> · 0.25 NaOH · 12 H <sub>2</sub> O.	1460
Potassium Manganate (VI) K <sub>2</sub> MnO <sub>4</sub>	1461
Barium Manganate (VII) $Ba(MnO_4)_2$	1462
Silver Manganate (VII) AgMnO <sub>4</sub> .	1463
BaSO <sub>4</sub> -KMnO <sub>4</sub> Solid Solution	1463
Potassium Manganese (III) Chloride K <sub>2</sub> MnCl <sub>5</sub>	1464
Potassium Hexachloromanganate (IV) K <sub>2</sub> MnCl <sub>2</sub>	1464
Manganese (II) Sulfide MnS.	1465
Manganese (III) Sulfate Mn <sub>o</sub> (SO <sub>4</sub> ) <sub>3</sub>	1467
Cesium Manganese (III) Sulfate $CsMn(SO_4)_2 \cdot 12 H_2O$ .	1468
Manganese Nitride Mn <sub>4</sub> N	1468
Manganese (III) Acetate Mn(CH <sub>3</sub> COO) <sub>3</sub> , Mn(CH <sub>3</sub> COO) <sub>3</sub> .	
2 H <sub>2</sub> O	1469

CON	TENT	s
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Potassium Trioxalatomanganate (III)	
$K_{3}[Mn(C_{2}O_{4})_{3}] \cdot 3 H_{2}O_{4}$	1470
Potassium Dioxalatodihydroxomanganate (IV)	
$K_{2}[Mn(C_{2}O_{4})_{2}(OH)_{2}] \cdot 2 H_{2}O_{4}$	1471
Potassium Hexacyanomanganate (I) K <sub>5</sub> Mn(CN) <sub>6</sub>	1472
Potassium Hexacyanomanganate (II) K <sub>4</sub> Mn(CN) <sub>e</sub> · 3 H <sub>2</sub> O	1473
Potassium Hexacyanomanganate (III) $K_3Mn(CN)_6$	1474
SECTION 26. RHENIUM	1476
Rhenium Metal	1476
Rhenium (III) Chloride ReCl <sub>3</sub>	1476
Rhenium (V) Chloride ReCl 5	1477
Potassium Rhenium (IV) Chloride K <sub>2</sub> ReCl <sub>6</sub>	1478
Rhenium (VI) Oxychloride ReOCl <sub>4</sub>	1479
Rhenium (VII) Oxychloride ReO <sub>3</sub> Cl	1480
Rhenium (IV) Oxide ReO <sub>2</sub>	1480
Rhenium (VI) Oxide $ReO_3$	1481
Rhenium (VII) Oxide $\operatorname{Re}_2O_7$	1482
Rhenium Rhenate (IV) Na <sub>2</sub> ReO <sub>3</sub>	1483
Ammonium Perrhenate NH <sub>4</sub> ReO <sub>4</sub>	1484
Barium Perrhenate Ba(ReO <sub>4</sub> ) <sub>2</sub>	1485
Barium Rhenate (VI) BaReO <sub>4</sub>	1485
Rhenium (IV) Sulfide ReS <sub>2</sub>	1486
Rhenium (VII) Sulfide Re $_{2}S_{7}$	1487
Barium Mesoperrhenate Ba <sub>3</sub> (ReO <sub>5</sub> ) <sub>2</sub>	1487
Workup of Rhenium Residues	1488
	1400
SECTION 27. 1RON	1490
	1490
Iron (II) Chloride FeCl <sub>2</sub> .	1491
from (III) Chloride $FeCI_3$	1492
Iron (II) Bromide FeBr <sub>2</sub>	1493
Iron (III) Bromide FeBr <sub>3</sub>	1494
$lron (ll) lodide Fel_2 \dots \dots$	1495
Iron (II) Oxide FeO.	1497
Iron (II) Hydroxide Fe(OH) <sub>2</sub>	1498
Iron (II, III) Oxide $Fe_3O_4$	1499
Iron (III) Hydroxide FeO(OH)	1499
Iron (III) Oxychloride FeOCI.	1501
Iron (II) Sulfide FeS	1502
Iron Nitrides $Fe_2N$ , $Fe_4N$	1502
Iron Carbide Fe <sub>3</sub> C	1503
Lithium Ferrate (III) LiFeO <sub>2</sub>	1504
Potassium Ferrate (VI) K <sub>2</sub> FeO <sub>4</sub>	1504
Potassium Iron (III) Sulfide KFeS <sub>2</sub>	1507
Basic Iron (III) Sulfate Fe <sub>3</sub> (SO <sub>4</sub> ) <sub>2</sub> (OH) <sub>5</sub> · 2 H <sub>2</sub> O or	
$3 \operatorname{Fe}_{2}O_{3} \cdot 4 \operatorname{SO}_{3} \cdot 9 \operatorname{H}_{2}O_{2}$	1507

Basic Iron (III) Acetate	
$[Fe_3(CH_3COO)_{c}(OH)_{2}]CH_3COO \cdot H_{2}O$	1508
Hexacvanoferric (II) Acid H <sub>4</sub> Fe(CN) <sub>6</sub>	1509
Ammonium Hexacvanoferrate (II) $(NH_{4})_{4}Fe(CN)_{6}$	1509
Hexacvanoferric (III) Acid HeFe(CN)	1510
Sodium Hexathiocyanoferrate (III) Na Fe(SCN).	
12 H_O	1511
Sodium Dentacyangamminoferrate (II)	1011
No. $[Fo(CN) - NH] - 3 H = 0$	1511
Sodium Dontoovanoamminofarmata (III)	-
No $[Fo(CN) NU]$ , $U = O$	1512
	1012
	1-10
SECTION 28. COBALT, NICKEL	1513
Metallic Cobalt	1513
Cobalt (II) Chloride CoCl <sub>2</sub>	1515
Hexaamminecobalt (II) Chloride $[Co(NH_3)_6]Cl_2$	1516
Cobalt (II) Bromide $CoBr_2$ , $CoBr_2 \cdot 6 H_2O$	1517
Cobalt (II) Iodide $\alpha$ -CoI <sub>2</sub> , $\beta$ -CoI <sub>2</sub> , CoI <sub>2</sub> · 6 H <sub>2</sub> O	1518
Cobalt (II) Oxide CoO	1519
Cobalt (II, III) Oxide $Co_3O_4$	1520
Cobalt (III) Hydroxide CoO(OH)	1520
Cobalt (II) Hydroxide Co(OH)2	1521
Cobalt Sulfides CoS, $CoS_2$ , $Co_3S_4$ , $Co_9S_8$	1523
Cobalt (III) Sulfate $Co_2(SO_4)_3 \cdot 18 H_2O \dots \dots$	1524
Cobalt Aluminate $CoAl_2O_4$	1525
Hexaammminecobalt (III) Nitrate $[Co(HN_3)_6](NO_3)_3$	1526
Cobalt (III) Amide $Co(NH_2)_3$	1526
Dicobalt Nitride Co <sub>2</sub> N	1529
Cobalt Nitride CoN.	1529
Cobalt Phosphides CoP <sub>3</sub> , CoP, Co <sub>2</sub> P	1530
Dicobalt Carbide Co <sub>2</sub> C	1531
Hexaamminecobalt (III) Chloride [Co(NH <sub>3</sub> ) <sub>6</sub> ]Cl <sub>3</sub>	1531
Chloropentaamminecobalt (III) Chloride [Co(NH <sub>3</sub> ) <sub>5</sub> Cl]Cl <sub>2</sub> .	1532
Nitropentaamminecobalt (III) Chloride	
$[Co(NH_2) \circ NO_2]Cl_2$	1534
Nitritopentaamminecobalt (III) Chloride	
$[C_0(NH_2)_5ONO]C]_2$	1535
Carbonatotetraamminecobalt (III) Sulfate	
$[C_0(NH_0), C_{O_0}] = SO_4 \cdot 3 H_0O_1$	1535
Dichlorotetraamminecohalt (III) Chloride	
	1536
Triethylenediaminecohalt (III) Bromide [Co en ]Bro	1538
Decaammine-u-peroxocobalt (III) Cobalt (IV) Sulfate	-000
$[(NH_{\star}) - CoIII(O_{\star})CoIV(NH_{\star}) - 1(SO_{\star}) - SO_{\star}H - 3 H_{\star}O_{\star}$	1540
Sodium Hexanitritocohaltate (III) Na $[Co(NO_{2})]$	1541
Datassium Herrovanocohaltate (III) Kageouno 2/6 ····	1541
TOTASSIUM MERACYAMOCOVALIATE (M) AS[OU(ON)6]	TTOT

Hexacyanocobaltic (III) Acid	1542
Metallic Nickel	1543
Nickel (II) Chloride NiCl <sub>2</sub>	1544
Hexaamminenickel (II) Chloride [Ni(NH <sub>3</sub> ) <sub>6</sub> ]Cl <sub>2</sub>	1545
Nickel (II) Bromide NiBr <sub>2</sub>	1545
Nickel (II) Iodide Nil <sub>2</sub>	1547
Nickel (II) Oxide NiO	1548
Nickel (II) Hydroxide Ni(OH) 2	1549
<b>B</b> -Nickel (III) Hydroxide NiO(OH)	1549
y-Nickel (III) Hydroxide NiO(OH)	1550
Nickel (II, III) Hydroxide Ni <sub>3</sub> O <sub>2</sub> (OH) <sub>4</sub>	1551
Nickel (II) Sulfide NiS	1551
Nickel (IV) Sulfide NiS2	1554
Nickel (II) Amide Ni(NH <sub>2</sub> ) <sub>2</sub>	1554
Trinickel Dinitride Ni <sub>3</sub> N <sub>2</sub>	1555
Trinickel Nitride NiaN	1555
Nickel Carbide Ni <sub>3</sub> C.	1556
Nickel (II) Carbonate	1556
Nickel (II) Thiocvanate Ni(SCN)	1558
$Di-\mu$ -sulfido-tetrakis(dithiobenzoato)dinickel (IV)	
$(C_{eH_{5}} \cdot CSS) = NiS = Ni(SSC \cdot C_{eH_{5}}) = \dots$	1558
Potassium Tetracyanonickelate (II)	
$K_2[Ni(CN)_4] \cdot H_2O_2$	1559
SECTION 29. THE PLATINUM METALS	1560
Pure Platinum Pt	1560
Reclaimed Platinum	1561
Platinum Sponge	1562
Platinum Black	1562
Platinized Asbestos	1563
Handling of Platinum Equipment	1564
Platinum Electroplating	1565
Platinum Chlorides	1567
Hexachloroplatinic (IV) Acid $H_2PtCl_6 \cdot 6 H_2O$	1569
Tetrachloroplatinic (II) Acid H <sub>2</sub> PtCl <sub>4</sub>	1570
Ammonium Hexachloroplatinate (IV) $(NH_4)_2$ PtCl <sub>6</sub>	1570
Potassium Hexachloroplatinate (IV) K <sub>2</sub> PtCl <sub>6</sub>	1571
Sodium Hexachloroplatinate (IV) Na <sub>2</sub> PtCl <sub>6</sub> ,	
$Na_2PtCl_6 \cdot 6 H_2O$	1571
Potassium Tetrachloroplatinate (II) K <sub>2</sub> PtCl <sub>4</sub>	1572
Platinum (II) Oxide PtO.	1573
Platinum (IV) Oxide $PtO_2 \cdot xH_2O$	1574
Hexahydroxyplatinates (IV) Na $_{2}$ Pt(OH) <sub>6</sub> · xH $_{2}$ O,	
$K_2$ Pt(OH) <sub>6</sub> • x H <sub>2</sub> O	1575
Platinum (II) Sulfide PtS	1575
#### CONTENTS

Potassium Tetracyanoplatinate (II) and Barium Tetra-	
cyanoplatinate (II) $K_2Pt(CN)_4 \cdot 3H_2O$ , $BaPt(CN)_4 \cdot$	
4 H <sub>2</sub> O	1576
Ammine Complexes of Platinum (II) (Platinum Ammines)	
$[Pt(NH_3)_4][PtCl_4]$ and $[Pt(NH_3)_4]Cl_2 \cdot H_2O \dots \dots$	1577
Reiset's Second Chloride trans-[PtCl <sub>2</sub> (NH <sub>3</sub> ) <sub>2</sub> ]	1578
Peyrone's Chloride cis-[PtCl <sub>2</sub> (NH <sub>3</sub> ) <sub>2</sub> ]	1578
cis-Dinitrodiammineplatinum (II) [Pt(NO <sub>2</sub> ) <sub>2</sub> (NH <sub>3</sub> ) <sub>2</sub> ]	1579
Pure Palladium Pd	1580
Colloidal Palladium	1581
Palladium Black.	1581
Palladized Asbestos	1582
Palladium (II) Chloride PdCl <sub>2</sub>	1582
PdCl <sub>2</sub> Solution for the Detection of CO	1582
Palladium (II) Oxide PdO.	1583
Tetrachloropalladates (II) K <sub>2</sub> PdCl <sub>4</sub> , Na <sub>2</sub> PdCl <sub>4</sub> ,	
$(NH_{4})$ $_{2}PdCl_{4}$	1584
Hexachloropalladates (IV) K <sub>2</sub> PdCl <sub>6</sub> , (NH <sub>4</sub> ) <sub>2</sub> PdCl <sub>6</sub>	1584
Diamminepalladium (II) Salts [PdCl_(NH <sub>3</sub> )].	
$[PdBr_{2}(NH_{3})_{2}]$	1585
Pure Rhodium Rh.	1585
Rhodium (III) Chloride RhCl <sub>3</sub>	1587
Hexachlororhodates (III)	1587
Rhodium (III) Oxide Rh <sub>2</sub> O <sub>3</sub>	1588
Rhodium Sulfate	1589
Chloropentaamminerhodium Salts [RhCl (NH <sub>3</sub> ) <sub>E</sub> ]Cl 2	
$[RhC1(NH_3)_5](NO_3)_2$	1590
Pure Iridium Ir	1590
Iridium (IV) Oxide IrO <sub>2</sub>	1590
Hydrated Iridium (IV) Oxide $IrO_2 \cdot 2 H_2O_1$	1591
Hydrated Iridium (III) Oxide $Ir_2O_3 \cdot x H_2O \dots$	1592
Iridium (III) Chloride IrCl <sub>3</sub>	1592
Hexachloroiridic (IV) Acid H <sub>2</sub> IrCl <sub>6</sub>	1593
Potassium Hexachloroiridate (IV) K <sub>2</sub> IrCl <sub>6</sub>	1593
Ammonium Hexachloroiridate (IV) (NH <sub>4</sub> ) <sub>2</sub> IrCl <sub>6</sub>	1594
Potassium Hexachloroiridate (III) K <sub>3</sub> IrCl <sub>6</sub> · 3 H <sub>2</sub> O	1595
Pure Ruthenium Ru	1595
Ruthenium (IV) Hydroxychloride Ru(OH)Cl <sub>3</sub>	1597
Ruthenium (III) Chloride RuCl <sub>3</sub> , RuCl <sub>3</sub> $\cdot$ H <sub>2</sub> O	1597
Ammonium Hexachlororuthenate (IV) $(NH_4)_2RuCl_6$	1599
Ruthenium (IV) Oxide RuO <sub>2</sub>	1599
Ruthenium (VIII) Oxide $RuO_4$	1599
Potassium Ruthenate and Potassium Perruthenate	
$K_2 RuO_4 \cdot H_2O$ , $KRuO_4$	1600
Pure Osmium Os	1601
Osmium (IV) Chloride OsCl <sub>4</sub>	1601
Sodium Hexachloroosmate (IV) Na <sub>2</sub> OsCl <sub>6</sub> · 2 H <sub>2</sub> O	1602

#### CONTENTS

Ammonium Hexachloroosmate (IV) (NH <sub>4</sub> ) <sub>2</sub> OsCl <sub>6</sub>	1603
Osmium (IV) Oxide OsO <sub>2</sub>	1603
Osmium (VIII) Oxide $OsO_4$	1603
Potassium Osmate (VI) K <sub>2</sub> OsO <sub>4</sub> · 2 H <sub>2</sub> O	1604
Potassium Osmiamate K(OsO <sub>3</sub> N).	1605

## Part III

# **Special Compounds**

SECTION 1. ADSORBENTS AND CATALYSTS	1609
Introduction	1609
Active Metals	1613
Pyrophoric Cobalt	1615
Ni-Mg Mixed Oxalate Catalyst (1:1)	1615
Tungsten	1622
"Molecular" Silver	1623
Raney Nickel	1625
Nickel Formate-Paraffin Catalyst	1631
Active Copper	1633
Carbonyl Iron	1636
Explosive Antimony	1638
Silver (Active Agent for Reductors)	1641
Deposition of Metals from the Vapor Phase	1643
Hydrated Oxide Gels.	1646
Hydrated Chromium Oxide Gel	1648
Silica Gel	1648
Aluminum Hydroxide Gel	1652
"Glimmering" Hydrated Iron (III) Oxide	1654
Active Metal Oxides	1656
Aluminum Oxide	1660
$\alpha$ -Iron (III) Oxide	1661
Magnesium Oxide	1663
Zinc Oxide	1664
Lead (IV) Oxide	1668
Colloidal Suspensions of Oxides in Gases (Smokes)	1669
Copper-Chromium Oxide	1672
Hopkalite (Hopcalite).	1674
• • • • • • • • • • • • • • • • • • • •	
SECTION 2. HYDROXO SALTS	1677
General	1677
Handling of Concentrated Alkali Hydroxides	1679
Sodium Hydroxozincates	1681
Sodium Tetrahydroxomagnesate Na <sub>2</sub> [Mg(OH) <sub>4</sub> ]	1683
Sodium Tetrahydroxocuprate (II) Na [Cu(OH)]	1684
Barium Hexahydroxocuprate (II) Ba <sub>2</sub> [Cu(OH) <sub>6</sub> ]	1685
Sodium Tetrahydroxoferrate (II) Na <sub>2</sub> [Fe(OH) <sub>4</sub> ]	1686

Strontium Hexahydroxonickelate (II) Sr <sub>2</sub> [Ni(OH) <sub>6</sub> ]	1686
Sodium Trihydroxostannate (II) Na[Sn(OH) <sub>3</sub> ]	1687
Sodium Hexahydroxochromate (III) $Na_3[Cr(OH)_6]$	1688
Sodium Hydroxoferrates (III)	1689
Barium Hydroxoferrates (III)	1690
Alkali Aluminates.	1692
Sodium Hexahydroxostannate (IV) Na [Sn(OH)]	1694
Sodium Hexahydroxoplumbate (IV) Na [Pb(OH)]	1694
Barium Oxohydroxostannate (II) Ba[Sn <sub>2</sub> O(OH) <sub>4</sub> ]	1696
SECTION 3. ISO- AND HETEROPOLY ACIDS AND THEIR SALTS	1698
Introduction.	1698
General Methods	1700
Isopoly Compounds	1702
Isopolyvanadates	1702
Isopolyniobates	1705
Isopolytantalates	1707
Isopolyarsenates	1709
Isopolymolybdates	1710
Isopolytungstates	1712
Isopolysulfates	1714
Heterpoly Compounds	1716
12-Tungstic Acid-1-Borates	1716
12-Tungstic Acid-1-Silicates	1717
10-Tungstic Acid-1-Silicates	1719
12-Tungstic Acid-1-Phosphates	1720
22-Tungstic Acid-2-Phosphates	1722
21-Tungstic Acid-2-Phosphates	1722
18-Tungstic Acid-2-Phosphates	1723
12-Tungstic Acid-1-Arsenates	1724
18-Tungstic Acid-2-Arsenates	1725
6-Tungstic Acid-1-Tellurates.	1726
Metatungstates, Dodecatungstates	1727
12-Molvbdic Acid-1-Silicates	1729
12-Molvbdic Acid-1-Phosphates	1730
18-Molvbdic Acid-2-Phosphates	1732
12-Molvbdic Acid-1-Arsenates	1734
18-Molvbdic Acid-2-Arsenates	1734
6-Molvbdic Acid-2-Arsenates	1736
12-Molvbdic Acid-2-Chromites	1737
6-Molvbdic Acid-1-Periodates	1738
48-Vanadic Acid-2-Phosphates and 24-Vanadic Acid-2-	
Phosphates	1739
SECTION 4. CARBONYL AND NITROSYL COMPOUNDS	1741
General Information	1741
Chromium, Molybdenum, Tungsten Carbonyls Cr(CO).	
$Mo(CO)_6$ , $W(CO)_6$	1741

Iron Pentacarbonyl Fe(CO) <sub>5</sub>	1743
Diiron Nonacarbonyl Fe <sub>2</sub> (CO) <sub>9</sub>	1744
Triiron Dodecacarbonyl $[Fe(CO)_4]_3$ or $Fe_3(CO)_{12}$	1745
Cobalt Carbonyls $[Co(CO)_4]_2$ , $[Co(CO)_3]_4$	1746
Nickel Carbonyl Ni(CO) <sub>4</sub>	1747
Dipyridine Chromium Tetracarbonyl, Tripyridine Chro-	
mium Tricarbonyl $Cr(CO)_4$ py <sub>2</sub> , $Cr(CO)_3$ py <sub>3</sub>	1749
o-Phenanthroline Nickel Dicarbonyl Ni(CO) <sub>2</sub> C <sub>12</sub> H <sub>8</sub> N <sub>2</sub> .	1750
Iron Tetracarbonyl Halides $Fe(CO)_4X_2$	1751
Iron Tetracarbonyl Dihydride Fe(CO) <sub>4</sub> H <sub>2</sub>	1752
Cobalt Tetracarbonyl Hydride Co(CO) <sub>4</sub> H	1753
Iron Carbonyl Mercury Fe(CO) <sub>4</sub> Hg	1755
Cobalt Carbonyl Mercury [Co(CO) <sub>4</sub> ] <sub>2</sub> Hg	1755
Ethylenediamine Iron Carbonyl [Fe en <sub>3</sub> ] [Fe $_2(CO)_8$ ]	1756
Pyridine Iron Carbonyl [Fe py 6] $[Fe_4(CO)_{13}]$	1758
Potassium Nitrosyl Tricarbonyl Ferrate [Fe(CO) <sub>3</sub> NO]K	1759
Iron Dinitrosyl Dicarbonyl Fe(NO) <sub>2</sub> (CO) <sub>2</sub>	1760
Cobalt Nitrosyl Tricarbonyl Co(NO)(CO) <sub>3</sub>	1761
Dinitrosyl Cobalt Halides (NO) <sub>2</sub> CoCl, (NO) <sub>2</sub> CoBr,	
(NO) <sub>2</sub> CoI	1761
Sodium Dinitrosyl Thioferrate Na[(NO) <sub>2</sub> FeS] • 4 H <sub>2</sub> O	1763
Ammonium Heptanitrosyl Trithiotetraferrate	
$\mathrm{NH}_{4}[(\mathrm{NO})_{7}\mathrm{Fe}_{4}\mathrm{S}_{3}] \cdot \mathrm{H}_{2}\mathrm{O}$	1764
Ethyl Dinitrosyl Thioferrate [(NO)2FeSC2H5]2	1765
Potassium Dinitrosyl Thiosulfatoferrate	
$K[(NO)_{2}FeS_{2}O_{3}] \cdot H_{2}O$	1766
Potassium Nitrosyl Cyanomolybdate	
$K_4[(NO)Mo(CN)_5] \cdot H_2O$	1766
Potassium Nitrosyl Cyanomanganate $K_3[(NO)Mn(CN)_5]$ .	1767
Sodium Nitrosyl Cyanoferrate Na <sub>2</sub> [(NO)Fe(CN) 5] • 2 H <sub>2</sub> O	1768
Sodium Carbonyl Cyanoferrate $Na_3[(CO)Fe(CN)_5]$	1769
SECTION 5 ALLOYS AND INTERMETALLUC COMPOUNDS	1771
General Bemarks	1771
Purity of the Starting Materials	1772
Form of the Starting Materials	1773
Preparation of Starting Mixtures.	1773
Crucible and Ampoule Methods	1774
Heating and Cooling	1782
Allov Synthesis under Pressure	1784
Melting Without a Container	1786
Comminution in the Absence of Air	1786
Distillation Method.	1789
Residue Methods	1791
Special Processes	1793
Silicides	1795
Borides	1798

#### CONTENTS

Amalgams	1801 1808 1808
Formula Index	1811
SUBJECT INDEX	1828
INDEX OF PROCEDURES, MATERIALS AND DEVICES	1855
Errata for Volume 1	1859

## SECTION 19

## Copper, Silver, Gold

## O. GLEMSER AND H. SAUER

## Copper (Pure Metal)

 $\begin{array}{rcl} \text{CuO} &+ & \text{H}_2 &= & \text{Cu} &+ & \text{H}_2\text{O} \\ \hline 79.5 & & 22.4 \ l. & & 63.5 & & 18.0 \end{array}$ 

A solution of electrolytic copper in 30% nitric acid is evaporated to dryness. The resultant nitrate is converted to the oxide by heating for 15 hours in an electrical furnace at 850°C. The oxide is then reduced at low temperature (250-300°C). The product is finely divided metallic copper.

Alternate method: Reduction of copper oxalate with hydrogen [K. Fischbeck and O. Dorner, Z. anorg. allg. Chem. <u>182</u>, 228 (1928)]. For preparative directions, see subsection on CuS, p. 1018.

PROPERTIES:

Atomic weight 63.54; m.p. 1084°C, b.p. 2595°C;  $d_4^{2\circ}$  8.93. Crystal structure: type A1.

REFERENCE:

H. Haraldsen. Z. anorg. allg. Chem. 240, 339 (1939).

## Colloidal Copper

An ammoniacal solution of  $CuSO_4$  (1:1000) is treated with a dilute solution of hydrazine hydrate (1:2000) in the presence of acacia (gum arabic). The hydrosol obtained upon heating is immediately poured into a parchment paper bag which has been presoaked in water for some time; it is dialyzed against water for four days.

PROPERTIES:

The hydrosol is copper-red under incident light and blue under transmitted light. If protected from air, it is stable for a limited time. REFERENCE:

A. Gutbier and G. Hofmeier. Z. anorg. allg. Chem. 44, 227 (1905).

## **Copper Hydride**

## CuH

I.

 $4 \operatorname{CuI} + \operatorname{LiAlH}_{4} = \operatorname{LiI} + \operatorname{All}_{8} + 4 \operatorname{CuH}_{761.8}$ 761.8 38.0 133.9 407.7 258.3

A pyridine solution of CuI is made to react at room temperature with a solution of  $\text{LiAlH}_4$  in ether-pyridine (the latter being prepared by mixing a concentrated ether solution of  $\text{LiAlH}_4$  with absolute pyridine), yielding a blood-red pyridine solution of CuH. The mixture is allowed to stand at room temperature for 4-6 hours to complete the reaction. The All<sub>3</sub> co-product is sparingly soluble in pyridine and precipitates to a large extent. It is then readily separated from the clear supernatant liquor by centrifugation. The residual All<sub>3</sub> and the soluble LiI are separated from the CuH by addition of an at least equal volume of ether to the pyridine solution. The resultant red-brown precipitate of CuH is separated by centrifugation, washed with ether, dissolved in pyridine, and reprecipitated with ether. This purification procedure is repeated twice. The ether is then evaporated in a high vacuum.

The reaction may also be carried out by treating a solution of CuI in pyridine-tetrahydrofuran-ether with an ether solution of lithium aluminum hydride. In this case, CuH precipitates as soon as the two solutions are mixed, while both AlI<sub>3</sub> and LiI remain in solution. The CuH precipitate is then purified as above (by dissolving in pyridine and reprecipitating with ether).

# II. PREPARATION OF COPPER HYDRIDE BY REDUCTION OF SOLUTIONS OF COPPER SALTS WITH HYPOPHOSPHOROUS ACID

A 65°C mixture of 25 g. of  $CuSO_4 \cdot 5 H_2O$  in 100 ml. of water and 20 ml. of 2N  $H_2SO_4$  is added to a solution of 21 g. of  $H_3PO_4$  in 300 ml. of water. After standing for 24 hours, the resultant precipitate is filtered and washed successively with water, alcohol and ether. Although the precipitation is not quantitative under these conditions, the product is relatively pure.

Small amounts of iron salt or halogen ion impurity interfere with the precipitation.

#### PROPERTIES:

Formula weight 64.55. Light red-brown color. Anhydrous when obtained by method I. Undecomposed (metastable) up to about  $60^{\circ}$ C; decomposes into the elements above this temperature, and

1004

rapidly at 100°C. Quite stable in 0°C water; just as in the thermal decomposition, dissociates into metallic copper and  $H_2$  from 45°C on, rapidly at 65°C. Dark red pyridine solution. Crystal structure: type B4 (expanded Cu lattice). Heat of formation: 5.1 kcal./mole.

**REFERENCES:** 

- I. E. Wiberg and W. Henle. Z. Naturforsch. 7b, 250 (1952).
- II. O. Neunhoeffer and F. Nerdel. J. prakt. Chem. <u>144</u>, 63 (1935);
   G. F. Hüttig and F. Brodkorb. Z. anorg. allg. Chem. <u>153</u>, 235, 242 (1926).

## Copper (I) Chloride

## CuCI

 $2 CuSO_4 + 2 NaCl + SO_2 + 2 H_2O = 2 CuCl + Na_2SO_4 + 2 H_2SO_4$ (5 H<sub>2</sub>O)

499.4 116.9 22.4 *l*. 36.0 198.0 142.1 196.2

Gaseous  $SO_2$  is bubbled through an aequous solution of 50 g. of  $CuSO_4 \cdot 5 H_2O$  and 24 g. of NaCl at 60-70°C until CuCl ceases to precipitate. The product is suction-filtered and washed with sulfurous acid, then with glacial acetic acid until the latter becomes colorless. The moist product is placed in a shallow dish or on a large watch glass and heated on a water bath until the odor of acetic acid is no longer detectable. It is stored in a tightly closed container.

Alternate methods: a) Acetyl chloride is added in drops to a boiling solution of cupric acetate in glacial acetic acid containing at least 50% of acetic anhydride by volume. When the color changes to yellow, the addition is stopped and the mixture is refluxed for 15 minutes. The resultant white solid is suction-filtered, washed with acetic anhydride, and dried at 140-150°C (D. Hardt, private communication).

b) Cupric chloride is heated to 150-200°C in glycerol. The CuCl obtained is filtered, washed with alcohol, and dried in vacuum [B. K. Vaidya, Nature (London) <u>123</u>, 414 (1928)].

c) Reduction of  $CuCl_2 \cdot 2$  H<sub>2</sub>O in a Na<sub>2</sub>SO<sub>3</sub> solution (R. N. Keller and H. D. Wycoff in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 1).

d) A solution of crystalline  $CuCl_2$  in hydrochloric acid is reduced over copper with exclusion of air (use a Bunsen valve; for a description of the valve, see Hackh's Chemical Dictionary, 3rd ed., the Blakiston Co., Phila.-Toronto, 1944). The product is poured into water [M. Denigés, Compt. Rend. Hebd. Séances Acad. Sci. <u>108</u>, 567 (1889)].

e) A mixture consisting of 1 part of  $CuSO_4 \cdot 5 H_2O$ , 2 parts of NaCl and 1 part of Cu turnings is heated (use a Bunsen valve) with

10 parts of  $H_2O$  until the color disappears completely. The mixture is poured into water, and CuCl crystallizes out [M. Denigés, Comptes Rendus Hebd. Séances Acad. Sci. <u>108</u>, 567 (1889)].

f) Sublimation of commercially pure CuCl over copper in a stream of HCl and argon at 900°C [J. B. Wagner and C. Wagner, J. Chem. Physics <u>26</u>, 1597 (1957)].

SYNONYM:

Cuprous chloride.

PROPERTIES:

Formula weight 99.00. White crystalline material. M.p.  $432^{\circ}$ C, b.p.  $1490^{\circ}$ C;  $d_{4}^{25}$  4.14,  $d_{4}^{422}$  3.677. Sparingly soluble in water  $(25^{\circ}$ C): 1.53 g./100 g. (partial decomposition in water: 2CuCl = Cu + CuCl<sub>2</sub>). Forms a green basic chloride in air. Soluble in hot conc. hydrochloric acid, conc. alkali chloride solutions, conc. aqueous ammonia. Crystal structure: type B3. Conversion into high-temperature modification of type B4 at 410°C. Heat of formation (25°C): - 32.2 kcal./mole.

**REFERENCE:** 

M. Rosenfeld. Ber. dtsch. chem. Ges. 12, 954 (1879).

## Copper (I) Bromide

CuBr

$2  \mathrm{CuSO}_4$	+	2 KBr	+	$SO_2$	+	$2 H_2O$	=	2 CuBr	+	$2 H_2 SO_4$	+	$K_2SO_4$
(5 H <sub>2</sub> O)												
499.4		238.0		22.4 l.		36.0		286.9		196.2		174.3

Stoichiometric quantities of pure  $CuSO_4 \cdot 5 H_2O$  and KBr are dissolved in boiled distilled water and the solution is filtered through hard filter paper. It is then heated to a moderate temperature and a fast stream of pure  $SO_2$  is passed through, with stirring, for about two hours. The passage of gas is continued until the mixture has cooled completely; the CuBr precipitates in the form of fine yellowish-white crystals. The solid is filtered while carefully excluding all light, resuspended 5-7 times in boiled distilled water into which some  $SO_2$  is bubbled, and filtered again. The product is finally washed with  $SO_2$ -containing alcohol, followed by  $SO_2$ -containing ether. The salt is dried for 3-4 days over  $H_2SO_4$ and KOH in a hydrogen atmosphere, and then in vacuum.

Alternate methods: a) Acetyl bromide is added in drops to a boiling solution of cupric acetate in glacial acetic acid, containing at least 50% of acetic anhydride by volume, until the solution becomes light green and a pure white precipitate appears (D. Hardt, private communication). b) Another starting material consists of the mixture used in the preparation of ethyl bromide from alcohol,  $Br_2$  and red phosphorus. The mixture is filtered and an excess of  $CuSO_4 \cdot 5 H_2O$  is added to the clear solution. The dark green solution is brought to a boil; crystallization soon follows [D. B. Briggs, J. Chem. Soc. (London) 127, 496 (1925)].

c) Synthesis from the elements [J. B. Wagner and C. Wagner, J. Chem. Physics <u>26</u>, 1597 (1957)].

SYNONYM:

Cuprous bromide.

PROPERTIES:

Formula weight 143.46. Colorless crystals. M.p. 498°C, b.p. 1345°C;  $d_4^{25}$  4.72. Insoluble in H<sub>2</sub>O; soluble in hydrogen halide solutions, nitric acid and aqueous ammonia. Heat of formation (25°C): - 24.9 kcal./mole.

CuBr exists in three modifications:  $\gamma$ -CuBr (type B3) below 391°C,  $\beta$ -CuBr (type B4) between 391 and 470°C,  $\alpha$ -CuBr (cubic) above 470°C.

REFERENCE:

J. N. Frers. Ber. dtsch. chem. Ges. 61, 377 (1928).

## Copper (I) Iodide

#### CuI

 $2 \operatorname{CuSO}_4 + 2 \operatorname{KI} + \operatorname{SO}_2 + 2 \operatorname{H}_2 O = 2 \operatorname{CuI} + 2 \operatorname{H}_2 \operatorname{SO}_4 + \operatorname{K}_2 \operatorname{SO}_4$ (5 H<sub>2</sub>O) 499.4 332.0 22.4 l. 36.0 380.9 196.2 174.3

The compound is obtained as a pure white solid by precipitation of a solution of  $CuSO_4 \cdot 5 H_2O$  with KI in the presence of a slight excess of sulfurous acid. The product is washed with water containing a small amount of  $SO_2$ , then (with exclusion of air) with pure alcohol, and finally with anhydrous ether. It is then filtered with suction and freed in vacuum of the last traces of ether. Residual strongly adhefing traces of water are best removed in a high vacuum, first at 110°C and finally somewhat above 400°C. A better product is obtained if a small quantity of iodine is added to the material after it has been dried at 110°C. This iodine is entirely removed at 400°C.

Alternate methods: a) Analogous to the preparation of CuBr from the reaction mixture used in the synthesis of ethyl iodide. Crystalline CuI is obtained [D. B. Briggs, J. Chem. Soc. (London) <u>127</u>, 496 (1925)].

b) Synthesis from the elements [J. B. Wagner and C. Wagner, J. Chem. Physics <u>26</u>, 1597 (1957)].

SYNONYM:

Cuprous iodide.

PROPERTIES:

Formula weight 190.45. Pure white crystalline powder. M.p.  $605^{\circ}$ C, b.p. 1336°C;  $d_4^{25}$  5.63. Quite stable in light and air, melts without decomposition in high vacuum and in a stream of oxygen-free N<sub>2</sub>. The solidified melt is clear and colorless (impure materials yield dark melts). Insoluble in H<sub>2</sub>O; soluble in acids and aqueous ammonia; soluble in alkali iodides. Heat of formation (25°C); -16.2 kcal./mole.

CuI exists in three modifications:  $\gamma$ -CuI (type B3) below 402°C,  $\beta$ -CuI between 402 and 440°C, and  $\alpha$ -CuI (cubic) above 440°C.

**REFERENCE**:

C. Tubandt, E. Rindtorff and W. Jost. Z. anorg. allg. Chem. <u>165</u>, 195 (1927).

## Copper (II) Chloride

#### CuCl<sub>2</sub>

### I. DEHYDRATION OF THE HYDRATE IN A STREAM OF HCl

Pure  $CuCl_2 \cdot 2 H_2O$  is recrystallized from dilute hydrochloric acid to remove traces of basic salt, and is then heated to constant weight at 140-150°C in a stream of dry HCl. The  $CuCl_2$  is stored in a desiccator over  $H_2SO_4$  and NaOH until all remaining traces of adhering HCl have been absorbed by the NaOH.

II.  $Cu(CH_3COO)_2 + 2 CH_3COCl = CuCl_2 + 2 (CH_3CO)_2O$ 181.6 157.0 134.5 204.2

A) CUPRIC ACETATE SOLUTION

Glacial acetic acid containing a small quantity of acetic anhydride is placed in the solvent flask of a Soxhlet extractor. The extraction section of the apparatus is filled with copper turnings, air is introduced, and the solvent is brought to a boil. The solution becomes saturated with copper acetate after 1-2 hours.

## B) ANHYDROUS CUPRIC CHLORIDE

The solution prepared in the Soxhlet via (A) is allowed to cool to  $35^{\circ}$ C, decanted from the solid which crystallizes out, and precipitated at 40-50°C with the stoichiometric quantity of acetyl chloride. Calculation of the stoichiometric quantity may be based on the solubility of cupric acetate in glacial acetic acid: 20 g./liter at  $35^{\circ}$ C. The precipitate is washed with either hot glacial acetic acid or cold acetic anhydride, both of which may be removed by a final washing with anhydrous ether. The product is dried at 120°C.

Alternate methods: a) High-vacuum dehydration of CuCl<sub>2</sub>·2H<sub>2</sub>O at 100°C [W. Biltz, Z. anorg. allg. Chem. <u>148</u>, 207 (1925)].

b) Refluxing of  $CuCl_2 \cdot 2 H_2O$  in SOCl<sub>2</sub>. Removal of the excess SOCl<sub>2</sub> by distillation and evaporation of residual solvent in vacuum [H. Hecht, Z. anorg. allg. Chem. <u>254</u>, 37 (1947)].

SYNONYM:

Cupric chloride.

#### PROPERTIES:

Formula weight 134.45. Yellow, deliquescent mass. M.p.  $630^{\circ}$ C, b.p.  $655^{\circ}$ C;  $d_4^{25}$  3.387. Soluble in H<sub>2</sub>O and alcohol. Solubility in ethyl alcohol (0°C) 31.9g.; in methyl alcohol (15.5°C) 67.8 g./100 ml. Soluble in acetone, yielding a dark green solution, which becomes yellow at high dilution. Heat of formation (25°C): - 49.2 kcal./mole.

**REFERENCES**:

- I. H. C. Jones and W. R. Veazey. Z. phys. Chem. 61, 654 (1908).
- II. D. Hardt. Z. anorg. allg. Chem. (in press); private communication.

## Copper (II) Bromide

CuBr<sub>2</sub>

I.

 $\begin{array}{rcl} CuO & + & 2 HBr = & CuBr_2 & + & H_2O \\ \hline 79.5 & 161.8 & 223.4 & 18.0 \end{array}$ 

The stoichiometric quantity of CuO [or Cu(OH)<sub>2</sub>] is dissolved in aqueous hydrobromic acid and the solution is evaporated in vacuum over  $H_2SO_{4*}$ 

п.

$$Cu(CH_{3}COO)_{2} \cdot H_{2}O + 3 CH_{3}COBr$$

$$199.6 368.9$$

$$= CuBr_{2} + 2 (CH_{3}CO)_{2}O + CH_{3}COOH + HBr$$

$$223.4 204.2 60.1 80.9$$

Finely divided Cu(CH<sub>3</sub>COO)<sub>2</sub>·H<sub>2</sub>O (4.0 g.) is placed in a Pyrex tube (18  $\times$  200 mm.) which is closed off with a rubber stopper

provided with a dropping funnel and a filtering tube. Agitation (magnetic stirrer) is started; 15 ml. of benzene is added, followed slowly by a 10% excess of  $CH_3COBr$ . The mixture is stirred for 30 minutes. The  $CuBr_2$  precipitate is allowed to settle and the supernatant is siphoned off through the filtering tube. The reaction is brought to completion by treating the residue with additional benzene and  $CH_3COBr$ . The supernatant liquid is removed by filtration and the  $CuBr_2$  is washed 3-4 times with anhydrous benzene. The product is dried at 150°C for two hours under nitrogen.

SYNONYM:

Cupric bromide.

PROPERTIES:

Black crystals, very deliquescent. M.p. 498°C, b.p. 900°C;  $d_4^{2\circ}$  4.710. Highly soluble in H<sub>2</sub>O, yielding a green solution; solubility (15°C) 122 g./100 g. H<sub>2</sub>O; soluble in acetone, alcohol and pyridine. Dry heating causes decomposition into CuBr and Br<sub>2</sub>. Evaporation of an aqueous solution also causes decomposition (at the b.p.). Depending on the temperature, CuBr<sub>2</sub> crystallizes from aqueous solutions with two or four molecules of water of crystallization, yielding highly deliquescent, brownish-green crystals. Crystal structure: monoclinic. Heat of formation (25°C): - 33.2 kcal./mole.

**REFERENCES:** 

- I. L. Vanino. Handbuch der präp. Chemie [Handbook of Preparative Chemistry], Part I, 2nd Ed., Stuttgart, 1921.
- II. G. W. Watt, P. S. Gentile and E. P. Helvenston. J. Amer. Chem. Soc. <u>77</u>, 2752 (1955).

## Copper Oxychloride

## CuCl<sub>2</sub> · Cu(OH)<sub>2</sub>

$2 \operatorname{CuCl}_2$	$+ CaCO_3$	+ H <sub>2</sub> O	$= \operatorname{CuCl}_2 \cdot \operatorname{Cu}(\operatorname{OH})_2$	$+ CaCl_2$	$+ CO_2$
268.9	100.1	18.0	232.0	111.0	22.4 l.

Stoichiometric quantities of cupric chloride, calcium carbonate (marble) and water are allowed to react in a bomb tube for 48 hours at 200°C. The product is filtered, freed from unreacted CuCl<sub>2</sub> by washing with boiling alcohol, and dried in a desiccator.

Alternate method: A conc. solution of  $CuCl_2$  is boiled for several hours with CuO. The liquid is decanted; the product is washed with acetone and dried [E. Hayek, Z. anorg. allg. Chem. <u>210</u>, 241 (1933)].

PROPERTIES:

Dark yellowish-green powder, decomposed by boiling water. Crystal structure: monoclinic.

REFERENCE:

G. Rousseau. Compt. Rend. Hebd. Séances Acad. Sci. <u>110</u>, 1262 (1890).

# Copper (I) Oxide

## Cu<sub>2</sub>O

I.

4 Cu <b>(CH</b> 3	COO) <sub>2</sub> +	$- N_2H_4$	+ 2 H <sub>2</sub> O
$(1 H_2)$	O)	$(1 H_2O)$	
798.	6	50.1	36.0
$= 2 \operatorname{Cu}_2$	$O + N_2$	+ 8 CH	I₃COOH
286.2	28.0	i i	480.4

A 20% hydrazine hydrate solution (3-5 ml.) is added to 50 ml. of concentrated copper acetate solution. The solution turns green, nitrogen evolves, and a yellow to orange precipitate of  $Cu_2O$  separates on standing. The product is washed with  $H_2O$ , followed by alcohol and ether. Care must be exercised to avoid an excess of hydrazine in the reduction, since such an excess causes reduction to metallic copper.

II.

$$4 Cu + O_2 = 2 Cu_2O$$
  
254.2 22.4 l. 286.2

Small copper plates (e.g., 5 mm.  $\times$  20 mm.  $\times$  10  $\mu$ ) are hung from platinum wires placed in a vertical tubular furnace; the latter is then heated to 1000°C in an atmosphere of technical grade N<sub>2</sub> (1% O<sub>2</sub>). While bringing to the desired temperature and cooling down, use only pure N<sub>2</sub>. The reaction is completed after about 24 hours. The product composition corresponds approximately to Cu<sub>2</sub>O [cf. C. Wagner and H. Hammen, Z. physik. Chem. <u>B40</u>, 197 (1938)].

Alternate methods: a) Equivalent amounts of CuO and Cu are heated in vacuum for five hours at 1000°C. The product is homogenized and reheated [F. W. Wrigge and K. Meisel, Z. anorg. allg. Chem. <u>203</u>, 312 (1932)].

b) Reduction of Fehling's solution with hydrazine sulfate [M. C. Neuburger, Z. Physik <u>67</u>, 846 (1931)].

c) Electrolysis of a weakly alkaline solution of NaCl at 80°C, using copper electrodes [B. B. Dey, A. Jorgarao, H. V. K. Udupa, S. Sampath and R. Viswanathan, J. Sci. Ind. Research (India) <u>13B</u>, 219 (1954); Hira Lal, J. Sci. Ind. Research (India) <u>12B</u>, 424 (1953)].

SYNONYM:

Cuprous oxide.

PROPERTIES:

Formula weight 143.08. Yellow powder. Red Cu<sub>2</sub>O is identical with the yellow variety, the difference in color being caused by particle size. M.p. 1232°C;  $d_4^{25}$  6.04. Insoluble in H<sub>2</sub>O; soluble in aqueous ammonia, conc. aqueous hydrogen halide solutions, markedly soluble in alkali hydroxides. Soluble in dilute oxyacids, with formation of Cu and Cu<sup>++</sup>. Crystal structure: type C3. Heat of formation (from 2 Cu +  $\frac{1}{2}$  O<sub>2</sub>): -40.0 kcal./mole (25°C).

REFERENCES:

- I. M. Straumanis and A. Cirulis. Z. anorg. allg. Chem. <u>224</u>, 110 (1935).
- II. H. Dünwald and C. Wagner. Z. phys. Chem. <u>B22</u>, 215 (1933); E. Engelhard, Ann. Phys. (V) <u>17</u>, 501 (1933).

## Copper (II) Oxide

#### CuO

The starting material, cupric nitrate, may be obtained by dissolving electrolytic copper in nitric acid and evaporating the solution to dryness on a steam bath:

$$2 \operatorname{Cu}(\operatorname{NO}_3)_2 = 2 \operatorname{CuO} + 4 \operatorname{NO}_2 + O_2$$
(3 H<sub>2</sub>O)
  
483.2 159.1 184.0 32.0

The cupric nitrate is dried in a drying oven, in which the temperature is raised very slowly from 90 to 120 °C. After the material has been completely converted to the green, loose basic salt (24 hr.), it is boiled with water and filtered. The dried salt is first heated slowly to 400 °C, resulting in removal of most of the nitric acid; it is then pulverized, slowly heated further to 850 °C, and maintained at this temperature for one hour. It is again ground to a fine powder, reheated for several hours to about 700 °C, and allowed to cool in a desiccator.

Alternate methods: a) Precipitation of  $Cu(OH)_2$  from a  $CuSO_4$  solution with ammonia, followed by calcination to CuO. The product is free of sulfate. Calcination temperature 600-700°C [A. A. Kazantsev, Khim. Zh., ser. B (Zh. Prikl. Khim.) <u>77</u>, 1108 (1938)].

b) Oxidation of very pure thin copper foil at  $1000^{\circ}$ C in a stream of pure O<sub>2</sub> [H. H. von Baumbach, H. Dünwald and C. Wagner, Z. phys. Chem. <u>B22</u>, 226 (1933); K. Hauffe and P. Kofstand, Z. Elektrochem. <u>59</u>, 399 (1955)].

c) Precipitation from  $CuCl_2 \cdot 4 H_2O$  with sodium hydroxide in the presence of  $CH_3OH$  [R. Fricke and J. Kubach, Z. Elektrochem. 53, 76 (1949)].

SYNONYM:

Cupric oxide.

PROPERTIES:

Formula weight 79.54. Black powder. M.p. 1336°C;  $d_4^{14}$  6.315. Soluble in acids and ammonia. After calcination at high temperatures, soluble only in boiling conc. acids. Crystal structure: type B26. Heat of formation (25°C): -37.1 kcal./mole.

REFERENCE:

R. Ruer and J. Kuschmann. Z. anorg. allg. Chem. 154, 69 (1926).

## Copper (II) Hydroxide

#### Cu(OH)₂

I. A solution of  $CuSO_4 \cdot 5 H_2O$  is treated at 70°C with 10% aqueous ammonia until a deep blue color appears. The solution is then allowed to react with the stoichiometric quantity of NaOH, yielding a precipitate which settles well. This is filtered, washed repeatedly with warm water, and dried in vacuum over conc.  $H_2SO_4$ .

II. Aqueous ammonia is added in drops to a boiling solution of  $CuSO_4 \cdot 5 H_2O$  until the initially green precipitate acquires a blue color. The crystalline basic sulfate thus obtained is filtered and carefully washed with water. It is then digested with a moderately concentrated NaOH solution, filtered, washed, and dried in vacuum over CaO or  $H_2SO_{4*}$ .

SYNONYM:

Cupric hydroxide.

PROPERTIES:

Light blue, crystalline powder. Insoluble in  $H_2O$ ; soluble in acids and aqueous ammonia; fairly soluble in concentrated NaOH. The crystalline form is stable at 100°C. Heating of the freshly

**REFERENCES:** 

- I. A. N. Agte and N. S. Golynko. Trudy Leningr. Khim.-Tekh. Inst. 8, 140 (1940).
- II. L. Vanino and E. Engert. Chemiker-Ztg. 48, 144 (1927);
   B. Röttger. J. prakt. Chem. <u>73</u>, 491 (1858); R. Fricke and J. Kubach. Z. Elektrochem. <u>53</u>, 76 (1949).

## Potassium Cuprate (III)

## KCuO<sub>2</sub>

A mixture of any available finely divided potassium oxide with CuO is heated to 400-500°C in carefully dried oxygen at 760 mm. Hg.

A) POTASSIUM OXIDE, KO<sub>x</sub>

The sealed (20 cm. long) glass tube d containing distilled potassium is placed in the constricted side tube b of the apparatus depicted in Fig. 276; the system is evacuated and flame-dried, then filled with dry nitrogen; tube a is fixed so that it slopes somewhat toward c. Tube d is raised above b and its lower end broken in a stream of nitrogen; it is then replaced in b by means of a wire attached to hook e. The system is again evacuated; the potassium inside b is melted and allowed to flow into a, care being taken to avoid plugging of the gas inlet.



Fig. 276. Preparation of finely divided potassium oxide.

Oxidation of the potassium is achieved by admixing increasing amounts of oxygen to the nitrogen. The quantity of  $O_2$  is adjusted by means of flowmeter f. When the reaction is complete, the resultant loose powder is homogenized in a vacuum ball mill (Fig. 55, p. 76) in a stream of dry  $N_2$ , and stored in sealed glass ampoules.

1014

ANALYSIS:

Potassium is determined as  $\text{KClO}_4$ ; the product may also be hydrolyzed and titrated as KOH.

B) REACTION OF KO<sub>x</sub> WITH CuO

The potassium oxide prepared in (A) is ground with the stoichiometric amount of CuO (K:Cu = 1:1) in the vacuum ball mill mentioned above. The grinding is carried out with careful exclusion of moisture, and is continued until the powder clings to the walls. This usually takes 5-20 minutes. The inner ground joint  $s_2$  of the ball mill is then connected to the outer joint  $s_2$  of the transfer device shown in Fig. 277 in such a manner that the Tshaped transfer piece is horizontal. It contains a movable aluminum pin n which fits fairly loosely into opening  $s_{3*}$ . To start with, s<sub>3</sub> is closed off with a ground cap. The mixture of oxides is transferred from the ball mill to the transfer device by shaking and knocking at the walls. The transfer device is disconnected from the mill in a stream of dry  $N_2$  and joint  $s_2$  is closed with a ground stopper; cap  $s_3$  is then removed and  $s_3$  is connected to joint  $s_3$  on the side tube *a* of the main apparatus of Fig. 277. A silicon carbide boat k is located exactly below  $s_3$ . Stopper  $s_2$  is removed, dry  $N_2$  is introduced through  $s_4$ , and a small portion of the material is pushed into the boat by raising and lowering the aluminum pin n. Careful shifting of the boat followed by repeated movement of the pin allows the boat to be filled completely.



Fig. 277. Charging and heating of the KO<sub>X</sub>-CuO mixture.

While the stream of dry N<sub>2</sub> continues to flow, the boat is shifted to reactor tube q, placed in furnace o (Fig. 277), which consists of an electrically heated quartz tube surrounded by a transparent protective tube. The mixture is heated to 450°C in very pure O<sub>2</sub>; the formation of KCuO<sub>2</sub> is complete after 24 hours.

PROPERTIES:

Formula weight 134.64. Crystalline powder, steel blue to deep blue. Decomposes vigorously in water, yielding a brown-black precipitate. Decomposes in dilute acids, evolving  $O_2$  and forming cupric salts. Evolves chlorine and oxygen in conc. hydrochloric acid. Decomposes with loss of weight on heating above 500°C in a stream of oxygen. Nonmagnetic; shows a characteristic x-ray diffraction pattern.

REFERENCE:

K. Wahl and W. Klemm. Z. anorg. allg. Chem. 270, 69 (1952).

## Schweizer's Reagent

I. Copper turnings are covered with 20% ammonia containing some  $NH_4Cl$ , and air is bubbled through the suspension. An azureblue solution of  $[Cu(NH_3)_4](OH)_2$  is formed. Evaporation of the solution in a stream of dry  $NH_3$  yields long, azure-blue needles of  $[Cu(NH_3)_4](OH)_2$ .

II. Freshly precipitated  $Cu(OH)_2$  is dissolved in 20% ammonia solution.

PROPERTIES:

Formula weight 165.68. Schweizer's reagent dissolves cellulose.

REFERENCE:

M. E. Schweizer. J. prakt. Chem. 72, 109, 344 (1857).

## Copper (I) Sulfide

Cu<sub>2</sub>S

$$2 \operatorname{Cu} + S = \operatorname{Cu}_2 S$$
  
127.1 32.1 159.1

I. A mixture of stoichiometric quantities of Cu and S is placed in a quartz tube, which is sealed in high vacuum. The tube is heated until the mixture melts.

II. An evacuated, sealed glass tube contains very pure Cu at one end, while the other is charged with the stoichiometric quantity of S purified by the method of von Wartenberg (p. 342). The reaction is complete after heating 1-2 days at  $400^{\circ}$ C.

Alternate methods: a) Cupric sulfide is heated in vacuum to the melting point of cuprous sulfide. The reaction is preferably

1016

carried out in a graphite crucible inside an evacuated tube [E. Posnjak, E. T. Allen and H. E. Merwin, Z. anorg. allg. Chem.  $\underline{94}$ , 95 (1916)].

b) Cupric sulfide obtained by precipitation from a  $CuSO_4$  solution with  $H_2S$  is reduced in a stream of  $H_2/H_2S$ . The optimum conditions are: a temperature of 700°C, a gas composition of 4.6%  $H_2$  and 95.4%  $H_2S$ , and a reaction time of one hour. The product is crystalline and quite pure [N. P. Diyev and E. M. Yakimets, Izv. Ural. Fil. Akad. Nauk SSSR <u>1955</u>, No. 3, 5; abstract in Chem. Abstr. 13,638a].

SYNONYM:

Cuprous sulfide.

PROPERTIES:

Blue to blue-black. M.p. 1127°C;  $d_4^{2\circ}$  5.6. Solubility (18°C): 4.95 × 10<sup>-5</sup> g./100 g. H<sub>2</sub>O. Very sparingly soluble in hydrochloric acid. Heat of formation (25°C): -19.6 kcal./mole.

Exists in two modification:  $\beta$ -Cu<sub>2</sub>S (hexagonal) below 91°C,  $\alpha$ -Cu<sub>2</sub>S (type C 1) above 91°C. The latter exists only with a copper deficiency, the composition being approximately Cu<sub>1.8</sub>S.

**REFERENCES:** 

- I. P. Rahlfs. Z. phys. Chem. (B) <u>31</u>, 157 (1936); P. Ramdohr. Z. prakt. Geol. <u>51</u>, 1 (1943).
- II. C. Wagner. Private communication.

Copper (II) Sulfide

CuS

$$Cu + S = CuS$$
  
63.5 32.1 95.6

The sulfide precipitated when cupric salt solutions are treated with  $H_2S$  is not uniform. A better product is obtained from the reaction of a solution of sulfur in CS<sub>2</sub> with pure copper powder obtained from copper oxalate. Copper from CuO is unsuitable; it strongly absorbs  $H_2O$  vapor and thus still contains some oxygen.

A) COPPER OXALATE

A solution of  $CuSO_4 \cdot 5 H_2O$  in water is reacted with an equal volume of conc.  $H_2SO_4$ . The solution is brought to a boil, and a slight excess of boiling aqueous oxalic acid is introduced in a thin stream. The crystalline, easily filtered oxalate is repeatedly

washed with pure water, filtered through a filter crucible, and further washed until no acid can be detected.

## B) COPPER POWDER

The copper oxalate is heated at  $130^{\circ}$ C to remove as much water of crystallization as possible. It is then placed in an electric furnace and heated to  $320^{\circ}$ C in a stream of purified H<sub>2</sub>. The decomposition starts suddenly and is accompanied by a rise in temperature. Heating is continued at  $220-260^{\circ}$ C, and the product is then allowed to cool (both operations are conducted under a stream of H<sub>2</sub>). The copper powder is stored under hydrogen.

C) COPPER SULFIDE

The copper powder obtained above is finely ground and covered with  $CS_2$  in a beaker. Somewhat more than the theoretical amount of S is dissolved in a large volume of  $CS_2$  and added to the contents of the beaker (the sulfur required may be obtained in sufficiently pure form by dissolving pure S in  $CS_2$ , filtering the solution through a glass filter crucible, and precipitating the filtrate with low-boiling petroleum ether). The resultant  $Cu_2S$  is transferred with the adhering  $CS_2$  to a bomb and covered with twice the amount of S required for the formation of CuS. The bomb is filled as completely as possible with  $CS_2$  and sealed. Then it is rotated along its long axis for four hours while surrounded with steam. The bomb is opened and contents filtered through a glass filter crucible and washed with  $CS_2$ ; the residual, adhering  $CS_2$ is removed in vacuum. The product is dried for 1-2 hours at 90-100°C in a vacuum of 0.1-1 mm. Hg.

SYNONYM:

Cupric sulfide.

PROPERTIES:

Black. M.p. (dec.) 200°C;  $d_4^{2\circ}$  4.6. Insoluble in H<sub>2</sub>O, alcohol and dilute acids. Solubility (18°C): 33.6 × 10<sup>-6</sup> g./100 g. H<sub>2</sub>O. Somewhat soluble in solutions of (NH<sub>4</sub>)<sub>2</sub>S and alkali polysulfides. Soluble without residue in KCN solution. Crystal structure: type B 18. Heat of formation (25°C): -12.1 kcal./mole.

**REFERENCE:** 

K. Fischbeck and O. Dorner. Z. anorg. allg. Chem. 182, 228 (1928).

## Copper (I) Selenide

#### Cu<sub>2</sub>Se

 $2 Cu + Se = Cu_2Se$ 127.1 79.0 206.0

Selenium vapor carried in a stream of nitrogen is passed over Cu placed in a porcelain boat. The Se is also in a porcelain boat located ahead of the Cu in the quartz reaction tube. A thermal gradient is obtained by means of two electric heaters which maintain the temperature of the Cu at about 400°C and that of the Se at about 300°C. A well-crystallized product is obtained.

Alternate methods: a) Heating a stoichiometric mixture of Cu and Se in an evacuated, sealed quartz tube [P. Rahlfs, Z. phys. Chem. (B) <u>31</u>, 1957 (1936)].

b) Preparation of  $Cu_2Se$  and CuSe from Cu and Se in a  $Cu_2SO_4$  solution [C. Goria, Gazz. Chim. Ital. <u>70</u>, 461 (1940)].

c) Passage of H<sub>2</sub>Se through solutions of Cu salts. Formation of CuSe and Cu<sub>2</sub>Se [L. Moser and K. Atynski, Mh. Chemie <u>45</u>, 235 (1925)].

d) Reduction of the basic selenite CuO · CuSeO<sub>3</sub> [W. Geilmann and F. R. Wrigge, Z. anorg. allg. Chem. <u>210</u>, 373 (1933)].

SYNONYM:

Cuprous selenide.

PROPERTIES:

Black.  $d_4^{21}$  6.84. Exists in two modifications: tetragonal  $\beta$ -Cu<sub>2</sub>Se (below 110°C), cubic  $\alpha$ -Cu<sub>2</sub>Se (above 110°C) (essentially a defect lattice deficient in copper). Heat of formation (25°C): -14.2 kcal./mole.

REFERENCES:

P. Rahlfs. Z. phys. Chem. (B) <u>31</u>, 157 (1936); W. Borchert. Z. Kristallogr. <u>106</u>, 5 (1945); G. Gattow and A. Schneider. Z. anorg. allg. Chem. <u>286</u>, 296 (1956).

## Copper (I) Telluride

Cu<sub>2</sub>Te

 $2 Cu + Te = Cu_2Te$ 127.1 127.6 254.7

Obtained by fusing electrolytic Cu with pure Te in a crucible under a protective layer of NaCl and KCl. SYNONYM:

Cuprous telluride.

PROPERTIES:

Gray-blue, brittle; homogeneous; contains 33 and 35 atom % Te. M.p. about 900°C.  $d_4^{25}$  7.338. Crystal structure: hexagonal (special type), defect lattice at Cu <<sub>2</sub> Te.

REFERENCE:

H. Nowotny. Z. Metallforsch. (Metallkunde) 1, 40 (1946).

## Copper (I) Sulfate

## $Cu_2SO_4$

 $2 Cu + 2 H_2 SO_4 = Cu_2 SO_4 + 2 H_2 O + SO_2$ 127.1 196.2 223.1 36.0

Copper turnings are placed in conc.  $H_2SO_4$  at a temperature of 200°C. The resultant green solution is added dropwise, through an asbestos filter, to an alcohol-ether mixture (1:1) or to methanol, causing Cu<sub>2</sub>SO<sub>4</sub> to precipitate in the form of almost white crystals. The product is decanted, washed with alcohol, and dried in vacuum. It cannot be prepared by treating CuCl or CuI with  $H_2SO_4$ .

Alternate method: Double decomposition of Cu<sub>2</sub>O with neutral dimethyl sulfate under anhydrous conditions [A. Recoura, Comptes Rendus Hebd. Séances Acad. Sci. 148, 1105 (1909)].

SYNONYM:

Cuprous sulfate.

**PROPERTIES:** 

Nearly white crystals or grayish powder. Decomposes in water to  $CuSO_4$  and Cu. Stable in dry air; decomposes slowly in moist air. Easily decomposed by heating; oxidizes at 200°C to CuO and  $CuSO_4$ . Heat of formation (25°C): -197.2 kcal./mole.

**REFERENCE:** 

J. G. F. Druce and G. Fowles. Chem. News 137, 385 (1928).

## Tetraamminecopper (II) Sulfate

### $[Cu(NH_3)_4]SO_4 \cdot H_2O$

A solution of 50 g. of finely divided  $CuSO_4 \cdot 5 H_2O$  in 75 ml. of conc. ammonia and 50 ml. of water is filtered and precipitated by slow addition of 75 ml. of alcohol. After standing for several hours in the cold, the crystals are filtered on a Büchner funnel, washed with a mixture of alcohol and conc. ammonia (1:1) and then with alcohol and ether, and dried by suction.

Large crystals may be obtained by covering a layer of alcohol with a layer of an ammonia solution of  $CuSO_4$  (G. Bornemann, Anorgan. Präparate [Inorganic Preparations], Leipzig, 1926, p. 156).

SYNONYM:

Cuprammonium sulfate.

PROPERTIES:

Deep blue crystals.  $d_4^{2^\circ}$  1.81. Solubility (21.5°C): 18.5 g./100 g. H<sub>2</sub>O. Decomposes in air. Loses H<sub>2</sub>O and 2 NH<sub>3</sub> on heating to 120°C; the remaining ammonia is evolved at 160°C.

REFERENCES:

H. and W. Biltz. Übungsbeispiele aus der anorg. Chemie [Exercises in Inorg. Chem.], Leipzig, 1920; F. Mazzi. Acta Cryst. <u>8</u>, 137 (1955); M. Simerská. Czechosl. J. Phys. <u>4</u>, 3 (1954).

## Copper (I) Nitride

#### $Cu_3N$

This compound is prepared by treating CuF<sub>2</sub> with NH<sub>3</sub>.

#### A) STARTING MATERIALS

1. According to L. Balbiano [Gazz. Chim. Ital.  $\underline{14}$ , 78 (1884)] CuF<sub>2</sub> · 2 H<sub>2</sub>O is prepared by dissolving CuO in 40% hydrofluoric acid, precipitating the fluoride with alcohol, and drying in a vacuum.

2. NH<sub>4</sub>F is dried in vacuum over NaOH.

3.  $NH_3$  and  $N_2$  are carefully dried.

B) DEHYDRATION OF  $CuF_2 \cdot 2H_2O$ 

About 1.5 g. of a mixture of 5 parts of  $CuF_2 \cdot 2H_2O$  and 1 part of  $NH_4F$ , in a corundum boat placed in a quartz tube set in an electric resistance furnace, is slowly heated for two hours to 280°C in a stream of N<sub>2</sub>. (The  $NH_4F$  serves to depress hydrolysis during the dehydration.)

C) PREPARATION OF Cu<sub>3</sub>N

The anhydrous  $CuF_2$ , at 280°C (see above), is immediately reacted for three hours at the same temperature with a fast stream of NH<sub>3</sub>. Heating above 300°C gives products deficient in nitrogen.

SYNONYM:

Cuprous nitride.

PROPERTIES:

Formula weight 204.63. Dark green powder, stable in air at room temperature; oxidizes at 400°C in a stream of  $O_2$  with pronounced incandescence. Decomposes spontaneously in vacuum at about 450°C. Soluble in dilute mineral acids and conc. hydrochloric acid with formation of the corresponding ammonium salt and partial formation of Cu metal. Decomposes violently with conc.  $H_2SO_4$  and  $HNO_3$ .  $d_4^{25}$  5.84. Crystal structure: type  $D0_9$ . Heat of formation (25°C): +17.8 kcal./mole.

REFERENCES:

R. Juza and H. Hahn. Z. anorg. allg. Chem. <u>239</u>, 282 (1938); <u>241</u>, 172 (1939); R. Juza. Ibid. <u>248</u>, 118 (1941).

Copper (II) Azide

 $Cu(N_3)_2$ 

 $\begin{array}{rcl} {\rm Cu(NO_3)_2} &+& 2\,{\rm NaN_3} &=& {\rm Cu(N_3)_2} &+& 2\,{\rm NaNO_3} \\ && & & \\ {\rm (3\,H_2O)} \\ && & 241.6 & 130.0 & 147.6 & 170.0 \end{array}$ 

A solution of 5 g. of  $Cu(NO_3)_2 \cdot 3 H_2O$  in 200 ml. of  $H_2O$  is treated in the cold with 50 ml. of a solution containing 2.5 g. of NaN<sub>3</sub>. The resultant precipitate is suction-filtered and washed several times with cold water. The wet product is left to stand 24 hours in 50 ml. of a 2% solution of hydrazoic acid, suctionfiltered, washed with alcohol and ether, and dried at room temperature. The yield is 2.5 g. of azide in the form of a brownblack mass with a reddish shine.

1022

Alternatively, finely powdered basic  $\rm CuCO_3$  may be treated with an excess of 2% HN \_3, after which the workup is the same as above.

Alternate methods: a) Reaction between  $Cu(NO_3)_2 \cdot 3 H_2O$  and  $LiN_3 \cdot H_2O$  in alcoholic solution and decomposition of the  $Cu(N_3)_2 \cdot 2 NH_3$  [M. Straumanis and A. Cirulis, Z. anorg. allg. Chem. <u>251</u>, 315 (1943)].

b) Determination of azide nitrogen according to F. Feigl and E. Chargaff, Z. anal. Chem. 74, 376 (1928).

SYNONYM:

Cuprous azide.

PROPERTIES:

Black-brown powder or black-brown, opaque crystal needles, depending on the method of preparation. Very sparingly soluble in  $H_2O$  and organic solvents. Readily soluble in acids, including  $CH_3COOH$ , and in ammonia. Decomposes on heating in air into Cu and N<sub>2</sub>. Can be easily reduced to white CuN<sub>3</sub> in an aqueous solution of hydrazine. Crystal structure: orthorhombic.

*Explosive properties:* Harmless when moist, quite sensitive to rubbing when dry or moistened with ether. Explodes when placed in a flame. Six times stronger than  $Pb(N_3)_2$  and 450 times stronger than mercury fulminate when used as a detonator.

REFERENCE:

M. Straumanis and A. Cirulis. Z. anorg. allg. Chem. <u>251</u>, 315 (1943).

## Copper Phosphide

#### Cu<sub>3</sub>P

$$3 Cu + P = Cu_3 P$$
  
190.6 31.0 221.6

Stoichiometric amounts of Cu and red P are heated for 20 hours at  $640^{\circ}$ C in an evacuated, sealed Vycor glass tube. The reaction product is homogenized, melted in a sealed quartz tube, and heated for five hours at  $1000^{\circ}$ C.

#### PROPERTIES:

Silvery, shiny material with metallic appearance. Insoluble in nitric acid.  $d_4^{25}$  7.147. Crystal structure: hexagonal. Heat of formation (25°C): - 36.0 kcal./mole.

REFERENCE:

## H. Haraldsen. Z. anorg. allg. Chem. 240, 337 (1939).

## Copper Diphosphide

#### CuP<sub>2</sub>

 $Cu_3P + 5P = 3CuP_2$ 221.6 154.9 376.5

A mixture of  $Cu_3P$  with the calculated amount of red phosphorus is heated in a quartz tube for 24 hours at 600°C.

PROPERTIES:

Formula weight 125.49. Gray-black, grainy powder. Slowly dissolves in boiling nitric acid (1.2).  $d_4^{25}$  4.201. Heat of formation (25°C): -23.5 kcal./mole.

**REFERENCE**:

H. Haraldsen. Z. anorg. allg. Chem. 240, 337 (1939).

## **Basic Copper Carbonates**

 $CuCO_3 \cdot Cu(OH)_2$  (Green Cupric Carbonate)

 $2 \operatorname{Cu}(\operatorname{NO}_3)_2 + 2 \operatorname{Na}_2 \operatorname{CO}_3 + \operatorname{H}_2 \operatorname{O}_3 + 4 \operatorname{Na}_2 \operatorname{O}_3 + 4 \operatorname{Na}_2 \operatorname{O}_3 + 4 \operatorname{Na}_3 \operatorname{Cu}_3 \cdot \operatorname{Cu}(\operatorname{OH})_2 + \operatorname{CO}_2 + 4 \operatorname{Na}_3 \operatorname{NO}_3 + 221.2 + 44.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340.0 + 340$ 

An aqueous solution of  $Cu(NO_3)_2 \cdot 3 H_2O$  is allowed to react at room temperature with a solution containing the equivalent amount of sodium or potassium carbonate. The greenish blue, partially colloidal precipitate of varying composition that forms is gradually transformed under the mother liquor into crystalline  $CuCO_3 \cdot Cu(OH)_2$ . Instead of  $Cu(NO_3)_2 \cdot 3 H_2O$ ,  $Cu(CH_3COO)_2 \cdot H_2O$ or  $CuSO_4 \cdot 5 H_2O$  may be used.

Alternate methods: a) Precipitation of 100 ml. of 1N  $CuSO_4$  with 110 ml. of 1N  $Na_2CO_3$ , followed immediately by filtering, washing with warm water, and drying after standing for 24 hours [M. Gröger, Z. anorg. allg. Chem. <u>24</u>, 127 (1900)].

1024

b) Hot  $CuSO_4$  solution is precipitated with sodium hydroxide. The precipitate is decanted and washed until the solid is free of alkali. It is then dissolved in acetic acid, the solution is evaporated to dryness, and the residue is taken up in water and added to a 100°C solution containing 4/5 of the equivalent quantity of  $K_2CO_3$ . The supernatant is decanted and the precipitate is washed with hot water and dried [W. C. Reynolds, Proc. Chem. Soc. (London) <u>190</u>, 53 (1897/98)].

PROPERTIES:

Malachite-green powder, insoluble in  $H_2O$ , soluble in aqueous ammonia. On boiling in water, particularly when the latter contains alkali carbonate, deposits brown oxide. Stable to 150°C in the absence of alkali, decomposes at 220°C. Unstable toward  $H_2S$ .  $d_4^{25}$  3.85.

REFERENCE:

G. Bornemann. Anorg. Präparate [Inorganic Preparations], Leipzig, 1926, p. 156.

## 2 CuCO<sub>3</sub> · Cu(OH)<sub>2</sub> (Blue Cupric Carbonate)

A solution of copper nitrate is mixed with an excess of pieces of chalk, and the mixture is placed in a large-diameter tube of strong glass connected to a mercury manometer. The tube is then sealed. The azurite forms at room temperature when the liberated  $CO_2$  creates a pressure of 5-8 atm.

Alternate methods: a) From precipitated green basic copper carbonate under a  $CO_2$  pressure of 4 atm. The reaction is markedly accelerated by the addition of azurite [V. Auger, Comptes Rendus Hebd. Séances Acad. Sci. <u>158</u>, 944 (1914)].

b) A soluble copper salt is added in portions to a solution containing  $Na_2CO_3$ ,  $NaHCO_3$  and suspended blue copper carbonate. A new portion is added only after the previous one has been converted from the green basic carbonate to the blue [V. Auger, loc. cit.].

c) Formation from  $CuCO_3 \cdot Na_2CO_3 \cdot 3H_2O$  (from saturated NaHCO<sub>3</sub> solution and precipitated basic CuCO<sub>3</sub>) and moist CO<sub>2</sub> at 40 atm. [V. Auger, loc. cit.].

PROPERTIES:

Azure-blue crystalline powder. M.p. (dec.)  $220^{\circ}$ C;  $d_4^{2\circ}$  3.88. Insoluble in water. Converts to the green compound in humid air. Soluble in ammonium salt solutions. Crystal structure: orthorhombic.

**REFERENCE:** 

H. J. Debray. Comptes Rendus Hebd. Séances Acad. Sci. <u>49</u>, 218 (1859).

## Copper (I) Acetylide

#### $Cu_2C_2 \cdot H_2O$

2 Cu <b>C</b> l	+	$H_2C_2 =$	$= Cu_2C_2$	+	2 HCl
100.0		00 <i>(</i> ]	$(H_2O)$		<b>73</b> 0
198.0		22.4l.	169.1		72.9

Pure CuCl (10 g.) is added in vacuum to a solution of 30 g. of NH<sub>4</sub>Cl in 100 ml. of H<sub>2</sub>O; it dissolves after addition of 50 ml. of conc. ammonia. A solution of 20 g. of hydroxylammonium chloride in 100 ml. of  $H_2O$  is then added, and the entire mixture is diluted with 150 ml. of H<sub>2</sub>O. The solution becomes completely colorless after a few minutes. It is then siphoned into an evacuated vessel, and acetylene is introduced. The acetylene (from a steel cylinder) passes through a purification train consisting of sealed wash bottles equipped with fritted glass plates and filled (in succession) with  $HgCl_2$  solution, 2N NaOH, Cu (NO<sub>3</sub>)<sub>2</sub> in nitric acid and 2N H<sub>2</sub>SO<sub>4</sub>, followed by two wash bottles filled with 2% leuco-indigo carmine solution (made from indigo carmine and zinc dust), for the detection and absorption of O<sub>2</sub>, and a glass-bead trap for catching any entrained liquid droplets. Upon contact with the cuprous salt solution, acetylene produces a bright red, flocculent and very voluminous precipitate. The product is suction-filtered on a fritted-glass funnel and washed with boiled water and acetone, all operations being carried out in vacuum. After thorough suctiondrying, it is dried at 100°C (in high vacuum) in a drying pistol. The product contains about 95% Cu<sub>2</sub>C<sub>2</sub> · H<sub>2</sub>O and is stored in sealed ampoules filled in high vacuum.

#### PROPERTIES:

Brownish-red powder. Insoluble in  $H_2O$ , soluble in HCl and KCN solutions. On heating with HCl, moist, freshly prepared  $Cu_2C_2 \cdot H_2O$  decomposes into  $C_2H_2$  and CuCl (and a small amount of vinyl chloride). Oxidizes in air to  $Cu_2O$ , C and  $H_2O$ , the color changing to dark brown.

**REFERENCES:** 

R. Klement and E. Köddermann-Gros. Z. anorg. Chem. <u>254</u>, 201 (1947);
 L. Ilosvay. Ber. dtsch. chem. Ges. <u>32</u>, 2697 (1899).

#### Paris Green (Copper Acetoarsenite)

 $4 \text{ CuO} + 3 \text{ As}_2\text{O}_3 + 2 \text{ CH}_3\text{COOH} = 3 \text{ Cu}(\text{AsO}_2)_2 \cdot \text{Cu}(\text{CH}_3\text{COO})_2 + \text{H}_2\text{O}$ 318.2 593.5 120.1 1013.7 18.0

Cupric oxide is heated with 8% acetic acid,  $As_2O_3$  is added, and the mixture is refluxed for two hours. The product is allowed to cool for half an hour, filtered, washed and dried.

Alternate method: Dilute acetic acid is allowed to react with an excess of freshly precipitated  $Cu(OH)_2$  and the product is separated by filtration. Dilute acetic acid is added to a solution of  $As_2O_3$  in boiling NaOH until the color of phenolphthalein disappears. The hot solutions are mixed (mole ratio of  $CuO:As_2O_3$ = 4:3) and allowed to stand for several days [S. Avery, J. Amer. Chem. Soc. <u>28</u>, 1159 (1906)].

**PROPERTIES:** 

Emerald green, crystalline powder, stable to air and light. Insoluble in  $H_2O$ . Decomposes on prolonged heating in  $H_2O$ . Unstable in acids, bases and toward  $H_2S$ . Toxic.

**REFERENCE:** 

G. D. Luchinskiy and U. F. Churilkina. Zh. Prikladnoy Khim. <u>13</u>, 558 (1940).

## Fehling's Solution

SOLUTION 1

34.6 g. of  $CuSO_4 \cdot 5 H_2O$  dissolved in 500 ml. of  $H_2O$ .

SOLUTION 2

173 g. of crystalline potassium sodium tartrate (Rochelle salt) and 53 g. of NaOH are dissolved in  $H_2O$  and diluted to 500 ml.

Equal volumes of the two solutions are mixed before use.

PROPERTIES:

Deep blue solution, reduced to  $Cu_2O$  on heating with reducing agents (sugar test in urine).

**REFERENCE:** 

J. D'Ans and E. Lax. Taschenbuch für Chemiker und Physiker [Pocket Book for Chemists and Physicists], Berlin, 1943, p. 1779.

## Very Pure Silver

Crude silver (e.g., from silver residues, see p. 1029) is dissolved in conc. nitric acid, and the diluted solution is precipitated in the cold with a solution of very pure NaCl. The precipitate is washed several times with cold water and dissolved in freshly prepared ammonia solution. After several hours of standing, the solution is filtered. Silver chloride is precipitated from the filtrate with very pure nitric acid, washed with H<sub>2</sub>O until no further nitrate can be detected, and reduced in a silver dish with invert sugar\* and NaOH (from Na metal) at 60°C, with sucrose and NaOH, or with a boiling alkaline solution of formaldehyde (the formaldehyde should be distilled prior to the preparation; sugar solutions should be filtered through bone charcoal and recrystallized). The resultant silver slurry is filtered, carefully washed free of chloride ion with water, dried and melted down to small ingots over pure CaO. If the metal is heated no longer than absolutely necessary for melting, and if the resultant metal grains are cooled in a reducing flame, the silver obtained is quite pure; it contains about 0.001% S and traces of C, AgCl and O.

The silver obtained in this manner is further purified by electrolysis. The greater part of the grains is used as the anode (fritted glass finger filled with the Ag) against a cathode of pure silver (wire or sheet). The electrolyte is a 10% AgNO<sub>3</sub> solution prepared from the remaining melted silver and very pure nitric acid. The power supply lead is a strip of fine sheet silver whose upper surface is protected by a layer of asphalt or Bakelite lacquer. The electrolysis is carried out at a constant voltage of 1.39 v. across the terminals. The silver flakes depositing on the cathode are removed from time to time, carefully washed, dried and fused to ingots in a stream of pure hydrogen in a boat made of very pure CaO.

<sup>\*</sup>A 1:1 mixture of dextrose and levulose.

PROPERTIES:

Atomic weight 107.88. M.p. 960.5°C, b.p. 2170°C; d<sup>2°</sup><sub>4</sub>° 10.497. Crystal structure: type A 1.

**REFERENCES**:

O. Hönigschmid and R. Sachtleben. Z. anorg. allg. Chem. <u>195</u>, 207 (1931); Th. W. Richards and R. C. Wells. Z. anorg. allg. Chem. <u>47</u>, 56 (1905); O. Hönigschmid, E. Zintland M. Linhard. Z. anorg. allg. Chem. <u>136</u>, 263 (1924).

## Silver Powder

Pure AgCl is stirred with water, allowed to react with sodium hydroxide, and reduced with glucose, which is added to the boiling suspension in small portions. Samples are removed from time to time, filtered and carefully washed. If the Ag yields a completely clear solution on heating in chloride-free HNO<sub>3</sub>, the reduction is complete. The medium must be kept alkaline throughout the reaction; an excess of glucose should be avoided. The product is filtered, washed free of base, and dried at  $100^{\circ}C$ .

Alternate methods: a) Reduction of AgCl with aqueous formaldehyde [L. Vanino, Ber. dtsch. chem. Ges. <u>31</u>, 1764 (1898)].

b) Heating finely powdered  $Ag_2O$  to  $500^{\circ}C$  [F. Jirsa and J. Jelinek, Z. anorg. allg. Chem. <u>158</u>, 63 (1926)].

c) Reduction of Ag<sub>2</sub>O with  $H_2O_2$  and drying at 250°C [F. Jirsa and J. Jelinek, loc. cit.].

PROPERTIES:

Gray, crumbling powder. According to Vanino, the product of the reduction is a loose, black powder.

REFERENCE:

G. Bornemann. Anorg. Präparate [Inorganic Preparations], Leipzig, 1926, p. 161.

## Silver from Residues

I. The collected residues are allowed to react with hydrochloric acid (1:1). The precipitate is allowed to settle and the supernatant is siphoned off. The precipitate is washed free of iron by repeated decantation with hydrochloric acid and water, filtered with suction,

placed in a large porcelain dish, mixed with HCl (1:1), and reduced with Zn rods with stirring. After disappearance of the white particles of AgCl, the Ag slurry is washed free of acid and Zn with hot water and filtered. The washings should be tested for the presence of Zn. The resultant silver slurry can be processed further either to Ag or  $AgNO_3$  (when only pure AgCl is required, an ammonia solution of the salt may be reduced with 20% hydrazine hydrate).

a) The silver slurry is dried and fused with a small quantity of borax in a Hessian crucible. The fused Ag is made into granules by careful pouring into water.

b) The silver slurry is dissolved in nitric acid (1:1); the solution is filtered and evaporated in a porcelain dish on a steam bath until crystallization. The last traces of nitric acid are removed by drying in a vacuum or by fusion.

Alternate methods: The silver obtained from the reduction with Zn is dissolved in dilute nitric acid. The solution is filtered and AgCl is precipitated by addition of dilute hydrochloric acid to the hot solution. The precipitate is filtered, carefully washed with warm water, and dried. Then 20 parts of AgCl are mixed in a mortar with 10 parts of Na<sub>2</sub>CO<sub>3</sub> and 3 parts of KNO<sub>3</sub>. The mixture is placed in a red-hot Hessian crucible. The reduction proceeds according to 2 AgCl + Na<sub>2</sub>CO<sub>3</sub> = 2 Ag + 2 NaCl + CO<sub>2</sub> +  $\frac{1}{2}$  O<sub>2</sub>. The AgCl may also be added in portions to the mixture heated slightly above 960°C. A melt of Ag is formed immediately. The Ag ingot is cleaned by boiling in water containing sulfuric acid (G. Bornemann, Anorg. Präparate [Inorganic Preparations], Leipzig, 1926, p. 160).

REFERENCES:

Handbuch für das Eisenhüttenlaboratorium [Handbook for the Iron Works Laboratory], Vol. I, p. 317 (1939). F. Specht. Quantitative anorganische Analyse in der Technik [Quantitative Inorganic Analysis in Engineering], 1953.

## II. SILVER FROM PHOTOGRAPHIC SOLUTIONS

The photographic solution is made alkaline with ammonia and allowed to react with a slight excess of ammonium sulfide. The mixture is allowed to stand overnight and the supernatant liquid is siphoned off. The residue is suction-filtered and washed with water. The precipitate, after addition of a small amount of anhydrous borax, is placed in a Hessian crucible, dried and calcined at 960°C. The borax is leached out of the product with hot water.

Alternate methods: a) Zinc rods are placed in exhausted fixing solutions and allowed to stand for about one week with frequent

agitation. The precipitated Ag slurry is filtered off and cupellated with lead (H. Grubitsch, Anorg. präparative Chemie [Inorganic Preparative Chemistry], Vienna, 1950, p. 454; for description of cupellation, see also Hackh's Chemical Dictionary, 3rd ed., The Blakiston Co., Philadelphia-Toronto, 1944).

b) The pH of the solution is adjusted to 6.9-7.2 with soda, and  $CuSO_4$  or  $Al_2(SO_4)_3$  is added. Silver precipitates with the corresponding hydroxide (when the Ag is not present as AgCl, FeCl<sub>3</sub> is added). The voluminous precipitate is treated after 3-4 days with sulfuric acid of increasing concentration (up to 96%), which removes hydroxides, gelatin and other impurities and concentrates the Ag to 20-50% of the total. The Ag is fused in a crucible after adding some borax (U.S. Pat. 2,131,045).

REFERENCE:

Handbuch für das Eisenhüttenlaboratorium [Handbook for the Iron Works Laboratory], Vol I, p. 318 (1939).

## Silver Mirrors

Of the two methods given below, the first is best for flat surfaces, while the second is used for concave surfaces, such as vacuum jackets. With variations, however, they may also be used in other applications. Careful and thorough cleaning of the mirror surface (glass, quartz, porcelain, mica, plastic, etc.) is a necessary condition for any successful silvering effort.

## I. A) PREPARATION OF THE MIRROR SURFACE

The silver coating on old mirrors is dissolved with nitric acid and the surface is rinsed with water. The hands are scrubbed with soap, and the soap foam is transferred to the surface, which is then scrubbed for some time with the foam. Scrubbing is continued while the surface is rinsed, first with tap water, then with distilled water. In the end, the units surface must be perfectly wetted by the water. If any greasy, water-repellent area remains, the entire operation must be repeated. After rinsing, the piece to be mirrored is placed in a dish of distilled water.

If it is desired to polish the silver mirror after deposition, the surface is first covered with a thin paste prepared from equal parts of alcohol, ammonia and precipitated calcium carbonate, and the paste is rubbed in vigorously with cellulose pulp. This cleaning mixture may then be removed with some fresh cellulose.

### **B)** PREPARATION OF THE SOLUTION

The following are weighed into three clean test tubes: 1)  $5 \pm 0.1$  g. of AgNO<sub>3</sub>, 2)  $1.1 \pm 0.05$  g. of AgNO<sub>3</sub>, 3)  $0.9 \pm 0.05$  g. of Rochelle salt. The second and third of these compounds are dissolved in a few milliliters of distilled water.

The following are held in readiness: two dark brown bottles (500 ml.) with ground stoppers; three beakers or Erlenmeyer flasks with capacities of a) 800, b) 300, c) 100 ml.; a 20-ml. burette; a glass funnel.

#### C) SILVER SOLUTION

The 5 g. of  $AgNO_3$  (test tube 1) is dissolved in 50-100 ml. of distilled water in vessel b; one third of the solution is held in reserve in c. Ammonia is added (with vigorous swirling) from the burette to the larger portion in b until the resultant deep brown precipitate just dissolves. Some  $AgNO_3$  solution is then added from the reserve, ammonia is again added in drops, and the procedure is repeated until the reserve has been exhausted. The last solution added should be  $AgNO_3$ , and the mixture should be somewhat turbid. If the solution is clear at the end of the operation, a few crystals of  $AgNO_3$  are dissolved in distilled water and added to the mixture until turbidity sets in. The Ag solution is diluted to 500 ml. with distilled water and transferred without filtering to one of the brown bottles. Thus protected against light, it may be stored almost indefinitely.

#### D) REDUCING SOLUTION

The solution of Rochelle salt (test tube 3) is added to 500 ml. of distilled water in vessel a; the mixture is brought to a boil, and the AgNO<sub>3</sub> solution from test tube 2 is added, first in drops (the addition causes a boiling point elevation and thus a delay in boiling—do not add the AgNO<sub>3</sub> too fast!), then more rapidly. The resultant brown-colored turbidity gradually transforms into a greenish-gray precipitate. The solution is boiled over a small flame for six additional minutes, filtered and stored in the second brown bottle. In this bottle the solution is stable for several months.

#### E) SILVERING

Small mirrors are best made in a thoroughly cleaned crystallizing dish or a photographic developing tray. Larger mirrors are best prepared by creating a 5-cm.-high leakproof rim of paraffined paper at the edge of the surface, so that the surface itself serves as the bottom of the dish.

#### 1032

Equal volumes of the silver-containing and reducing solutions are measured out in a graduated cylinder in amounts sufficient to cover the mirror surface with a layer of liquid 1 cm. deep. The mixture is then immediately poured on to the surface and the dish is vigorously rocked. A bluish, rapidly thickening deposit is formed after a few minutes on the mirror and the glass walls. The solution becomes turbid and small silver particles appear on the surface of the liquid. The solution is poured off, the mirror rinsed with distilled water, and the silvering process repeated with fresh solution. Finally, the silvered piece is rinsed with distilled water followed by alcohol, and the mirror is allowed to dry while standing on end. It is advisable to grip the mirror with lab. tongs or forceps and not to touch it with the fingers.

In the case of mirrors silvered on the back, the silver layer is protected with a lacquer coating (shellac, varnish). The Ag precipitated on the glass side is removed with a cotton pad moistened with highly diluted nitric acid.

F) POLISHING

The operation is carried out on the day following the silvering. A piece of dust-free chamois leather is tightened around a ball of wool ("polishing ball"). The surface is then carefully gone over with the ball and is then rubbed with increasing pressure. If this does not produce the desired result, some jeweler's rouge is spread on the ball, the excess is brushed off, and polishing is continued.

The mirror prepared as above has a golden sheen (on the back side).

## II. A) PRETREATMENT

The glass surface to be treated, such as the inner space of a jacket, is cleaned for 30 minutes at  $60^{\circ}$ C with freshly prepared cleaning solution and is then thoroughly rinsed with water. This is followed by a 10-minute treatment with 1.4% hydrofluoric acid and another rinse with water. Then the surface is treated for 10 minutes with technical grade conc. nitric acid (d 1.52) and rinsed with water; the final rinse is distilled water.

B) SILVERING

The following three solutions are prepared separately:

1) 50 g. of  $AgNO_3$  in two liters of water; stored in the dark in a ground-joint bottle;

2) 90 g. of very pure, chlorine-free KOH in two liters of water; rubber-stoppered bottle;
3) 80 g. of sugar in 800 ml. of water; a mixture of 100 ml. of 96% alcohol and 3.5 ml. of very pure, chlorine-free nitric acid (d 1.42) is added to this solution, which is then stored for at least one week before use in a ground-joint bottle.

The three solutions (16:8:1) are mixed as follows: conc. ammonia is added in drops with stirring to solution (1) until the initially formed precipitate just dissolves. Solution (2) is then added, yielding a dark brown to black precipitate (a green precipitate indicates a deficiency of ammonia and the material must be rejected). If the precipitate is of the right color, ammonia is again added slowly with stirring until the precipitate just disappears. A slight excess of ammonia delays the deposition of silver in the next stage, but does not prevent it. The resultant mixture of (1) and (2) may be stored, at most, for one hour. Solution (3) is added immediately before use. The deposition of silver lasts 10 to 30 minutes, depending on the excess of ammonia. Its completion is recognizable by the appearance of a flocculent precipitate. The solution must be removed at this point to avoid harm to the mirror. The surface is thoroughly rinsed with distilled water to remove all residues (including silver slurry) and dried, preferably in vacuum. The silver layer and the glass do not separate on heating to  $450^{\circ}$ C in a high vacuum of  $10^{-5}$  mm., which is particularly important in silvering of vacuum jackets for distillation columns.

**REFERENCES**:

- I. W. Bothe. J. prakt. Chem. <u>42</u>, 191 (1863); R. Böttger. Polytechn. Notizbl. <u>38</u>, 217 (1883); <u>39</u>, 324 (1884); H. Kreusler, Die Sterne <u>9</u>, 42 (1929); E. von Angerer. Techn. Kunstgriffe [Industrial Techniques], 5th ed., Braunschweig, p. 61.
- II. P. W. Schenk. Private communication.

## Colloidal Silver

A mixture of 200 ml. of 30% FeSO<sub>4</sub> solution, 280 ml. of 40% sodium citrate, and about 50 ml. of 10% NaOH is added to 200 ml. of 10% AgNO<sub>3</sub> solution. The resultant colloidal silver precipitate is allowed to settle and washed 4-5 times with 10% ammonium nitrate solution and finally twice with 96% alcohol. The mixture is centrifuged and the product carefully dried on a water bath or in a desiccator.

Alternate methods: a) A 0.001N AgNO<sub>3</sub> solution (100 ml.) is treated with a few drops of freshly prepared tannin solution and one drop of 1% sodium carbonate solution. Heating the mixture results in formation of a sol (W. Ostwald, Kleines Praktikum der Kolloidchemie [Lab. Manual for Colloid Chem.], 7th ed., 1930, p. 4). b) A warm 0.001N solution of  $AgNO_3$  is reduced by dropwise addition of 0.005% hydrazine hydrate solution (Ag sol according to Gutbier, cited in W. Ostwald, loc. cit.).

c) Silver sol by electrical atomization: Two silver rods, 2-3 mm. in diameter, are bent at right angles 2 cm. from their ends and the bent sections immersed in a beaker with distilled water so that they form a U figure. A current of 4-6 amp. should flow at 110 volts through the short-circuited electrodes (rheostat control). Clouds of colloidal Ag are formed when an electric arc is passed through the gap between the two ends. Addition of a few drops of 2% sodium carbonate solution is recommended [G. Bredig, Angew. Chem. 951 (1898)].

PROPERTIES:

Black, grainy powder containing about 97% Ag. Soluble in water, yielding a red-brown to black, extremely finely divided Ag sol.

**REFERENCE:** 

M. Carey Lea. Amer. J. Sci. <u>37</u>, 476 (1889).

## Silver lodide

#### AgI

 $AgNO_3 + KI = AgI + KNO_3$ 169.9 166.0 234.8 101.1

At atmospheric pressure silver iodide exists in three modifications:  $\alpha$ -form (cubic, type B3),  $\beta$ -form (hexagonal, type B4) and  $\gamma$ -form (cubic, type B3). At room temperature, the rate of interconversion between the  $\alpha$ - and  $\beta$ -forms is so low that the two forms are stable when stored alone or as a mixture. The silver iodide precipitated from solutions usually consists of a mixture of these two modifications. When physical uniformity of the product is not a factor, chemically pure AgI may be prepared in large quantities by the following method.

Very pure KI (83 g.) is dissolved in 8.3 liters of distilled water, and 85 g. of very pure  $AgNO_3$  in 8.5 liters of water. The KI solution is added, with constant stirring, to the  $AgNO_3$  solution. A milky liquid is initally formed; flocculation occurs only later. The supernatant is siphoned off after standing 2-3 hours, the precipitate is transferred to a three-liter bottle, two liters of distilled water is added, and the mixture is shaken vigorously to disperse the small clumps of iodide. The flocculent precipitate settles rapidly, and the clear supernatant may be siphoned off after about five minutes. The product is cleaned by decantation until all the  $KNO_3$  has been removed. The wash water is allowed to remain in contact with the precipitate overnight to remove all of the possibly adsorbed electrolyte (which is hard to dislodge). (Test for  $KNO_3$ : evaporation to dryness of 200 ml. of the wash water in a platinum dish. Blank test with the distilled water used for washing.) The precipitate is placed on a piece of hard filter paper and dried at 110-120°C. The dry AgI is easily ground to a fine powder.

The mixing of the solutions and the washing operations must be carried out in the absence of daylight. The product may be exposed to daylight only when it is completely free of impurities.

PROPERTIES:

Yellow, crystalline. M.p. 556.8°C, b.p. 1506°C. Insoluble in  $H_2O$ ; solubility (25°C): 0.25  $\cdot$  10<sup>-6</sup> g./100 g.  $H_2O$ ; almost insoluble in ammonia; appreciably soluble in conc. hydriodic acid and conc. solutions of alkali iodides, particularly when hot; soluble in Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> solution. Heat of formation -14.95 kcal. per mole.

HEXAGONAL  $\beta$ -AgI

Silver iodide as precipitated above is dissolved in a conc. solution of potassium iodide. The solution is filtered and poured into water. The AgI precipitates as a thick flocculate. It is washed with water (by decantation) until the iodide ion is no longer detectable by the AgNO<sub>3</sub> test. The AgI is filtered and dried at room temperature.

d<sup>3°</sup> 5.696; crystal structure: type B4.

CUBIC  $\alpha$  - AgI

a) Cubic AgI is always formed when hexagonal or mixed AgI is pulverized.

b) Silver iodide is dissolved in a conc. solution of AgNO<sub>3</sub>. The solution is filtered and poured into water. Fine-grained, slowly coagulating AgI is formed. It is washed by decantation until the silver ion cannot be detected in the wash water. The mixture is filtered and the product dried at room temperature.

d<sup>30</sup> 5.680; crystal structure: type B3.

**REFERENCE**:

N. H. Kolkmeijer and J. W. A. van Hengel. Z. Kristallogr. <u>A 88</u>, 317 (1934).

## Silver Chlorate

#### AgClO<sub>3</sub>

Solutions of 170 g. of AgNO<sub>3</sub> and 106 g. of NaClO<sub>3</sub>, each dissolved in 100 ml. of 85°C water, are combined and cooled to 0°C. The supernatant is carefully decanted and 50 ml. of H<sub>2</sub>O (0°C) is added to the residue. The resultant crystals are suction-filtered; they are 95% pure.

A purer product may be obtained by dissolving the residue remaining after the above decantation in 125 ml. of  $H_2O$  at 90°C, cooling to 0°C, and suction-filtering. The crystals are redissolved in 120 ml. of  $H_2O$  at 90°C, cooled to 0°C, suction-filtered, and dried in a desiccator. The yield is about 118 g. of 99.7% AgClO<sub>3</sub>.

The compound should be stored in dark flasks. Since  $AgClO_3$  is a strong oxidant, extreme care should be exercised when it is brought into contact with easily oxidized materials, especially organic substances.

PROPERTIES:

White crystals. M.p. 230°C, decomposes at 270°C; d 4.43.

REFERENCE:

D. G. Nicholson and C. E. Holley in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 4.

## Silver Oxide

#### Ag<sub>2</sub>O

 $2 \text{ AgNO}_3 + 2 \text{ NaOH} = \text{Ag}_2 \text{O} + 2 \text{ NaNO}_3 + \text{H}_2 \text{O}$ 339.8 80.0 231.8 170.0 18.0

Equivalent quantities of a conc. solution of  $AgNO_3$  and a dilute solution of NaOH (both prepared with  $CO_2$ -free water) are mixed; the resultant precipitate is decanted and washed with  $CO_2$ -free water. The precipitate is centrifuged, suction-filtered, and dried at 85-88°C in a stream of  $CO_2$ -free air.

Alternate methods: a) Precipitation of a dilute  $AgNO_3$  solution with  $Ba(OH)_2$ , with careful exclusion of  $CO_2$  [E. Laue, Z. anorg. allg. Chem. 165, 336 (1927)].

b) Electrolytic preparation: A 25-30% NaNO<sub>3</sub> solution is electrolyzed at room temperature at minimum current density (anode current density is important), using a silver anode and a nickel cathode positioned as close as possible. [K. Merei, Magyar. Chem. Folyoirat <u>45</u>, 197 (1913)].

PROPERTIES:

Dark brown to brown-black powder. M.p. (dec.)  $300^{\circ}$ C;  $d_4^{25}$  7.220. Insoluble in water [solubility (20°C):  $2.14 \cdot 10^{-3}$  g./100 g. H<sub>2</sub>O]; somewhat soluble in NaOH. Moist Ag<sub>2</sub>O is quite insensitive to light. Some decomposition occurs on drying. Crystal structure: type C3. Heat of formation (25°C): -7.3 kcal./mole.

REFERENCE:

E. H. Madsen. Z. anorg. allg. Chem. 79, 197 (1913).

## Silver Peroxide

## $Ag_2O_2$

Ag <sub>2</sub> O	+	2  KOH	+	$2  \text{KMnO}_4$	=	$Ag_2O_2$	+	$2 K_2 MnO_4$	+	$H_2O$
231.8		112.2		316.1		247.8		394.2		18.0

Solutions of AgNO<sub>3</sub> and KMnO<sub>4</sub> are combined and an excess of KOH is added. The resultant precipitate is filtered in a glass filtering crucible, washed with ice-cold water until the filtrate is colorless, dried for two hours at 110°C, and placed for 24 hours in a desiccator over  $P_2O_5$ . An anhydrous product containing up to 60% Ag<sub>2</sub>O<sub>2</sub> is obtained.

Alternate methods: a) Treatment of AgNO<sub>3</sub> solution with a solution of potassium or ammonium persulfate [H. Marshall, J. Chem. Soc. (London) <u>59</u>, 775 (1891)].

b) Reaction of metallic Ag with ozone-containing  $O_2$  at 240°C; use of lower temperatures is also possible if  $Fe_2O_3$  or Pt is used as catalyst [W. Manchot and W. Kampschulte, Ber. dtsch. chem. Ges. <u>40</u>, 2891 (1907)].

c) Reaction of NaOCl with  $Ag_2O$  at 75-80°C [R. L. Dutta, J. Indian Chem. Soc. 32, 95 (1955)].

d) Hydrolysis of  $Ag_7NO_{11}$  (from anodic oxidation of  $AgNO_3$  solutions) and thermal decomposition of  $Ag_7NO_{11}$  [G. M. Schwab and G. Hartmann, Z. anorg. allg. Chem. 281, 183 (1955)].

PROPERTIES:

Gray-black powder. Decomposes above 100°C into Ag and  $O_2$ ;  $d_4^{25}$  7.483. Soluble in conc. HNO<sub>3</sub>, from which it precipitates

on dilution. Decomposes in hot conc.  $H_2SO_4$  with evolution of  $O_2$ . Strongly oxidizing.

REFERENCE:

F. Jirsa. Z. anorg. allg. Chem. 225, 302 (1935).

#### Sodium Orthoargentite

#### Na<sub>3</sub>AgO<sub>2</sub>

 $Ag_2O + 3Na_2O = 2Na_3AgO_2$ 231.8 186.0 417.7

Stoichiometric quantities of pure, absolutely dry Na<sub>2</sub>O and dry Ag<sub>2</sub>O (dried in vacuum at 80°C) are weighed in the absence of moisture and CO<sub>2</sub>. The two starting materials are ground to a fine powder and intimately mixed in the vacuum ball mill shown in Fig. 55, p. 76. The mixture is transferred (with exclusion of air) to a sintered magnesia boat placed in a protective tube of Fe or Ag which itself is positioned in a heating tube (Vycor or quartz). The heating tube is evacuated and the mixture heated to 400°C. The product is homogenized by grinding and reheating.

PROPERTIES:

Formula weight 208.85. Light green powder, highly sensitive to moisture. Decomposed to a black substance even by small quantities of water vapor. Stable up to 450°C in vacuum.

REFERENCE:

E. Zintl and W. Morawietz. Z. anorg. allg. Chem. 236, 372 (1938).

## Silver (I) Sulfide

#### $Ag_2S$

 $2 \text{ Ag} + \text{S} = \text{Ag}_2 \text{S}$ 215.8 32.1 247.8

I. Pure sulfur vapor, carried in a stream of  $N_2$ , is passed over pure Ag at 250 °C. The S and Ag are in two separate boats placed in a quartz tube. Alternatively the S may be placed in a quartz tube 18 cm. long and 16 mm. I.D. with a constriction in the middle (Fig. 278). That tube is open at one end and narrows to a conical nozzle at the other. The Ag is placed in a quartz tube, of the same diameter and about 9 cm. long, constricted at both ends in such a way that one of the ends fits the nozzle of the other tube. The other end is attached by means of a ground joint to an outlet tube for unreacted S vapor. The S, in a boat, is placed in the first of the two chambers formed by the 18-cm.-long quartz tube. Then the tubes containing the S and Ag are placed inside a larger quartz tube set in an electric furnace. The furnace is heated to  $350^{\circ}$ C in such a way that sulfur is distilled from the first into the second chamber of the 18-cm.-long tube (purification). The furnace is then set at  $250^{\circ}$ C and heating is continued in a stream of N<sub>2</sub>.



Fig. 278. Preparation of silver (I) sulfide.

Plugging of the outlet tube with excess S is prevented by heating the tube (if necessary, with an additional electric furnace). Depending on the flow rate, the conversion of 10 g. of Ag requires 6 to 8 hours. The excess S remaining in the Ag<sub>2</sub>S after completion of the reaction is removed by heating in a stream of N<sub>2</sub>. The temperature is maintained below 300°C to prevent sulfur from being driven off from the Ag<sub>2</sub>S.

II. Very pure Ag and S are placed at opposite ends of an evacuated, sealed glass tube previously cleaned by the von Wartenberg method (p. 342). An excess of S is used. The tube is heated for 1-2 days at 400°C. It is then pulled (halfway) out of the furnace so that the end containing the sulfur is exposed on the outside. This enables the free sulfur to sublime from the product onto the cold excess sulfur. The reaction is complete within about 12 hours.

Alternate methods: a) Precipitation of an ammoniacal solution of AgNO<sub>3</sub> with H<sub>2</sub>S. The precipitate is washed with water and dried at 150°C in a stream of CO<sub>2</sub>. Excess S is removed by heating for one hour at 350-400°C in a stream of H<sub>2</sub> [E. von Britzke and A. F. Kapustinski, Z. anorg. allg. Chem. <u>205</u>, 95 (1932)].

b) Fusion of the calculated amounts of Ag and S in a pressure vessel [A. M. Gaudin and D. W. McGlashan, Econ. Geol. <u>33</u>, 143 (1932)].

SYNONYM:

Argentous sulfide.

1040

**PROPERTIES**:

Black, crystalline. M.p. 845°C;  $d_4^{2\circ}$  7.234,  $d_4^{17^{6.4}}$  7.072. Soluble in KCN solution, insoluble in aqueous ammonia. Transition point  $\beta$ -Ag<sub>2</sub>S (rhombic)  $\rightarrow \alpha$ -Ag<sub>2</sub>S (cubic): 179°C. Heat of formation ( $\alpha$ -Ag<sub>2</sub>S, 25°C): -7.5 kcal./mole.

**REFERENCES:** 

- I. O. Hönigschmid and R. Sachtleben. Z. anorg. allg. Chem. <u>195</u>, 207 (1931).
- II. C. Wagner. Private communication.

## Silver (I) Selenide

#### Ag<sub>2</sub>Se

 $2 \text{ Ag} + \text{Se} = \text{Ag}_2 \text{Se}$ 215.8 79.0 294.7

A boat containing selenium and (behind it) a boat containing metallic silver are placed inside a quartz tube. The tube is surrounded by two electrical furnaces, so that the Se is heated to  $300^{\circ}$ C, while the Ag is at  $400^{\circ}$ C. A stream of O<sub>2</sub>-free nitrogen carries the Se to the silver (see Fig. 278, preparation of Ag<sub>2</sub>S). The Se vapor is passed over the metal until large amounts of Se begin to accumulate behind the Ag. The heat is maintained at  $400^{\circ}$ C for some time after that to remove any Se which may have adhered to the product surface. The conversion of 5 g. of Ag requires 6-8 hours. Dissociation begins above  $400^{\circ}$ C.

Alternate method: Reaction of soluble Ag salts with CuSe. The reactions with  $Cu_2Se$  and  $Cu_2Se_2$  give metallic Ag as a by-product [W. Geilmann and Fr. W. Wrigge, Z. anorg. allg. Chem. <u>210</u>, 378 (1933)].

PROPERTIES:

Black, crystalline.  $d_4^{24}$  8.187. Crystal structure: at 75°C tetragonal with c/a = 0.66; at 80°C tetragonal with c/a = 0.94; at 240°C cubic. Heat of formation (cubic Ag<sub>2</sub>Se, 25°C): -5.0 kcal. per mole.

REFERENCE:

O. Hönigschmid and W. Kapfenberger. Z. anorg. allg. Chem. 212, 198 (1933).

# Silver (I) Telluride

#### Ag<sub>2</sub>Te

 $2 \text{ Ag} + \text{Te} = \text{Ag}_2 \text{Te}$ 215.8 127.6 343.4

Prepared by passing Te vapor over Ag. Nitrogen serves as the carrier gas and the reaction temperature is 470°C. The flow rate of the Te vapor should be slow; i.e., the amount of Te entering the reaction chamber should not exceed the reactive capacity of the Ag. The apparatus is identical to that used for  $Ag_2Se$ . The conversion of 3 g. of Ag requires 72-96 hours. Excess Te is removed by heating to 500-540°C in high vacuum. Well-crystallized products are obtained.

Alternate method: Heating Ag in a porcelain crucible in an atmosphere of Te, with exclusion of air [P. Rahlfs, Z. phys. Chem. (B)  $\underline{31}$ , 157 (1936)].

**PROPERTIES:** 

Gray-black, crystalline.  $d_4^{2\circ}$  8.318. Transition point  $\beta$ -Ag<sub>2</sub>Te (rhombic)  $\neg \alpha$ -Ag<sub>2</sub>Te (cubic): 149.5°C. Heat of formation ( $\beta$ -Ag<sub>2</sub>Te, 25°C) -5.0 kcal./mole.

**REFERENCE:** 

O. Hönigschmid, Z. anorg. allg. Chem. 214, 281 (1933).

## Silver Sulfate

#### Ag<sub>2</sub>SO<sub>4</sub>

 $2 \text{ AgNO}_{3} + \text{ H}_{2}\text{SO}_{4} = \text{ Ag}_{2}\text{SO}_{4} + 2 \text{ HNO}_{3}$ 339.8 98.1 311.8 126.0

A solution of  $AgNO_3$  in some  $H_2O$  is treated with 1:1 sulfuric acid. The  $Ag_2SO_4$  precipitate is centrifuged, dissolved in hot conc.  $H_2SO_4$  (in a Pt dish if high product purity is required), and boiled for several minutes to expel the nitric acid. The acid sulfate which crystallizes on cooling is centrifuged and treated with water in a Pt dish. The normal sulfate thus crystallizes as a fine powder (evolution of heat). The supernatant is decanted and the crystals washed with pure water until free of acid. The  $Ag_2SO_4$ is centrifuged and dried on an air bath at 110°C. The entire operation must be conducted in a dustproof atmosphere.

Alternate methods: a) Treatment of AgNO<sub>3</sub> solution with Na<sub>2</sub>SO<sub>4</sub> solution [O. Hönigschmid and R. Sachtleben, Z. anorg. allg. Chem. 195, 207 (1931)].

b) Solution of Ag metal in sulfuric acid (O. Hönigschmid and R. Sachtleben, loc. cit.).

c) Finely divided  $Ag_2SO_4$  is obtained by precipitation of its aqueous solution with alcohol. The product is dried in vacuum at 100°C [H. Hahn and E. Gilbert, Z. anorg. allg. Chem. <u>258</u>, 91 (1949)].

PROPERTIES:

Colorless crystals. M.p. 657°C, decomposes at 1085°C;  $d_4^{15}$  5.460. Sparingly soluble in water; solubility (18°C): 2.57  $\cdot$  10<sup>-2</sup>. g./100 g. H<sub>2</sub>O. Slightly decomposed by light, acquiring a light-violet color. Dissociates on melting, acquiring a yellow color which disappears on treatment with gaseous SO<sub>3</sub>. Crystal structure: orthorhombic. Heat of formation (25°C): -170.5 kcal./mole.

REFERENCE:

Th. W. Richards and G. Jones. Z. anorg. allg. Chem. 55, 72 (1907).

## Silver Sulfite

#### Ag<sub>2</sub>SO<sub>3</sub>

 $2 \text{ AgNO}_3 + \text{Na}_2 \text{SO}_3 = \text{Ag}_2 \text{SO}_3 + 2 \text{ NaNO}_3$ 339.8 126.0 295.8 170.0

A solution of  $AgNO_3$  is treated at room temperature with the stoichiometric quantity of  $Na_2SO_3$  solution, yielding a precipitate of  $Ag_2SO_3$ , which is filtered, washed with well-boiled water, and dried in vacuum.

Alternate method: Precipitation of aqueous  $AgNO_3$  with sulfurous acid [J. Muspratt, Liebigs Ann. 50, 286 (1853)].

PROPERTIES:

Colorless powder. Sparingly soluble in water, soluble in aqueous ammonia. Decomposes in light and on heating, forming the dithionate and sulfate. Insoluble in liquid SO<sub>2</sub>.

REFERENCES:

P. Berthier. Ann. Chim. Phys. (3) <u>7</u>, 82 (1817); K. Seubert and M. Elten. Z. anorg. allg. Chem. <u>4</u>, 44 (1894).

## Silver Amide

#### AgNH<sub>2</sub>

 $\begin{array}{rcl} AgNO_3 \ + \ KNH_2 \ = \ AgNH_2 \ + \ KNO_3 \\ 169.9 & 55.1 & 123.9 & 101.1 \end{array}$ 

Silver amide is formed in the reaction between potassium amide and silver nitrate, both dissolved in liquid ammonia. Good results are obtained with the two-arm tube shown in Fig. 279, in which  $\text{KNH}_2$  solution is allowed to flow into excess  $\text{AgNO}_3$  solution, causing precipitation of  $\text{AgNH}_2$ .

A) KNH<sub>2</sub>

First,  $KNH_2$  is prepared in arm B from potassium metal and  $NH_3$ . To achieve this, inlet tube c is connected to a source of absolutely dry NH<sub>3</sub> and the entire apparatus is dried by passing through  $NH_3$  and heating. Then *a* is temporarily closed off with a stopper and oxide-free K, together with a few milligrams of sponge Pt as catalyst, is introduced at b in a stream of NH<sub>3</sub>. Inlet tube b is melt-sealed to a pressure-resistant tip, the pressure gradient required for the glass-blowing operation being achieved by brief alternate closing and opening of inlet a with a finger. The  $AgNO_3$  is now introduced into arm A and dried in a stream of  $NH_3$ . Arm A is then melt-sealed at a in the same way as b, the pressure gradient being achieved by removing the plug from stopcock h and closing the resultant two openings (when required) with a finger. Arm B is then immersed in ice water and  $NH_3$  is allowed to condense in it until it fills 1/4 of the volume. In the presence of Pt, 1 g. of K is converted to KNH<sub>2</sub> within 15 minutes. After completion of the reaction, the H<sub>2</sub> by-product is allowed to escape via stopcock h.



Fig. 279. Preparation of silver amide.

B) AgNH<sub>2</sub>

The AgNO<sub>3</sub> in arm A is dissolved in a manner similar to that presented above, and both arms are then filled to half their volume with liquid NH<sub>3</sub>. After the solutions have become homogeneous, they are combined by allowing the KNH<sub>2</sub> solution to flow into the AgNO<sub>3</sub> solution. The resultant precipitate of AgNH<sub>2</sub> settles well and is purified by decantation with liquid NH<sub>3</sub>. This is carried out by pouring the supernatant liquid NH<sub>3</sub> into arm B and redistilling it into A (B is then in lukewarm water, A in ice water). The precipitate is vigorously agitated, together with the NH<sub>3</sub> which condenses on it, and is then allowed to settle. The supernatant liquid NH<sub>3</sub> is again poured off into *B*. This operation is repeated several times. Finally a deep layer of liquid NH<sub>3</sub> is condensed onto the precipitate in *A*. This arm is then immersed in a -35 °C bath in order to establish atmospheric pressure inside the apparatus, stopcock *h* is opened, and the tube is fused at *d*. The liquid NH<sub>3</sub> is allowed to evaporate slowly through stopcock *h* and the tube is evacuated to remove residual NH<sub>3</sub>.

These reactions may also very conveniently be carried out in the apparatus of Fig. 69, p. 87; this apparatus is an improved version of the one described above.

**PROPERTIES:** 

White, quite voluminous precipitate (when moist). Soluble in ammonium salt solutions, absolute ammonia, and excess  $KNH_2$ ; insoluble in excess  $AgNO_3$ . Blackens on exposure to light. The precipitate shrinks considerably on drying and acquires color. Extraordinarily explosive when dry. Apparently impossible to obtain an absolutely pure state.

**REFERENCES:** 

E. C. Franklin. Z. anorg. allg. Chem. <u>46</u>, 1 (1905); R. Juza. Z. anorg. allg. Chem. <u>231</u>, 121 (1937).

## Silver Azide

#### AgN<sub>3</sub>

A solution of NaN<sub>3</sub> is treated in the cold with a slight excess of AgNO<sub>3</sub> solution. The AgN<sub>3</sub> precipitate is decanted, filtered, washed with water, alcohol and ether, and dried in vacuum over conc.  $H_2SO_4$ .

Alternate methods: a) Slow addition of a 1% AgNO<sub>3</sub> solution to 1.5% aqueous  $HN_3$ , prepared by distillation of a solution of  $NaN_3$  in  $H_2SO_4$ . The product is washed free of Ag ion [F. V. Friedländer, J. Amer. Chem. Soc. <u>40</u>, 1945 (1918); T. Curtius, Ber. dtsch. chem. Ges. <u>23</u>, 3027 (1890)].

b) Precipitation of a cold, saturated solution of AgNO<sub>3</sub> with hydrazine sulfate [A. Angeli, Atti Acad. dei Linc. 2, 569 (1893)].

PROPERTIES:

Colorless crystalline needles. M.p. 252°C. Sparingly soluble in water and nitric acid; readily soluble in aqueous ammonia. **REFERENCE**:

G. Tammann and C. Kröger. Z. anorg. allg. Chem. <u>169</u>, 16 (1928).

## Silver Nitride

#### Ag<sub>3</sub>N

Potassium hydroxide pellets are added to a solution of AgCl in conc. ammonia until the effervescence, caused by the evolving  $NH_3$ , stops. The mixture is diluted with distilled water, filtered through filter paper, and washed with water until the filtrate is neutral. The moist product is transferred from the filter paper to a porcelain dish, where it may be stored under water for some time.

The product contains small amounts of AgCl and Ag, but is free of  $Ag_2O$ .

Alternate methods: a) A solution of  $Ag_2O$  in conc. aqueous ammonia is allowed to stand in air or heated on a water bath. The same may be achieved by precipitation with alcohol. The product is impure, with a variable content of  $Ag_2O$  and Ag [F. Raschig, Liebigs Ann. 233, 93 (1886)].

b) Solid AgF  $\cdot$  2NH<sub>3</sub> is stored for several days over H<sub>2</sub>SO<sub>4</sub> in a desiccator. The product is free of Ag<sub>2</sub>O and rich in Ag; the yield is small [L. J. Olmer and Dervin, Bull. Soc. Chim. France (4) 35, 152 (1924)].

SYNONYM:

(Berthollet's) fulminating or detonating silver.

PROPERTIES:

Formula weight 337.65. Black flakes, sometimes shining black; crystalline appearance (when prepared according to Raschig, see above). Insoluble in  $H_2O$ , soluble in dilute mineral acids, explosive reaction with conc. acids. Both the dry and the moist product may be stored in air at room temperature for a long time. Slowly decomposes at 25°C. Decomposes at room temperature in vacuum. Decomposes explosively in air at about 165°C. Very sensitive

1046

(explodes) when touched with objects of great relative hardness, even when moist. Extremely sensitive when dry, but relatively easy to handle when moist. Explodes readily when prepared by Raschig's method.  $d_4^{19}$  9.0. Crystal structure: cubic. Heat of formation (25°C): +61.0 kcal./mole.

**REFERENCE:** 

H. Hahn and E. Gilbert. Z. anorg. Chem. 258, 77 (1949).

## Silver Acetylide

#### $Ag_2C_2$

Pure acetylene is introduced into a solution of  $AgNO_3$  which has been treated with a large excess of ammonia. The white precipitate of  $Ag_2C_2$  is filtered, washed with water, then with alcohol and ether, and dried over  $P_2O_5$  in a desiccator.

PROPERTIES:

White powder, light sensitive, very explosive, particularly when dry. Soluble (decomposition) in KCN solution, yielding  $C_2H_2$ . Decomposed by HCl into AgCl and  $C_2H_2$ . Decomposes hydrolytically in water and alkalis.

REFERENCE:

J. Eggert. Z. Elektrochem. 24, 150 (1918).

#### Silver Cyanamide

#### $Ag_2CN_2$

Careful addition of HNO<sub>3</sub> to commercial CaCN<sub>2</sub> at 0°C and pH 6 yields  $H_2CN_2$ . The solution thus obtained, which contains about 10%  $H_2CN_2$ , is treated with an ammoniacal solution of AgNO<sub>3</sub> (added in small portions). The resultant Ag<sub>2</sub>CN<sub>2</sub> is purified by solution in dilute HNO<sub>3</sub> and reprecipitation with dilute NH<sub>3</sub>; it is washed and rapidly dried at 130°C.

PROPERTIES:

Formula weight 255.79. Yellowish-white powder containing 99.5%  $Ag_2CN_2$ . Soluble in KCN:  $Ag_2CN_2 + 4$  KCN = 2K[Ag(CN)\_2] + K\_2CN\_2. Pyrolysis in vacuum proceeds via the intermediate

silver dicyanamide,  $AgC_{2}N_{3}$ , which is stable up to 600°C. The decomposition is complete at 750°C. The residue consists of Ag which is free of cyanide and N<sub>2</sub>. If the temperature rise is too rapid, the pyrolysis becomes explosive.

REFERENCE:

A. Chrétien and B. Woringer. Compt. Rend. Hebd. Séances Acad. Sci. <u>232</u>, 1114 (1951).

## Silver Carbonate

#### Ag<sub>2</sub>CO<sub>3</sub>

A solution of  $AgNO_3$  is treated with alkali carbonate or bicarbonate. When precipitating with the carbonate, avoid an excess of the reagent, since the  $Ag_2CO_3$  precipitate may then contain oxides. The product is filtered, washed with water, and dried to constant weight over  $H_2SO_4$  and  $P_2O_5$ . It still contains traces of water.

Due to its sensitivity to light, a pure silver carbonate can be obtained only when the preparation is carried out in red light.

Alternate method: Electrolysis of a 0.02M solution of NaHCO<sub>3</sub> using a silver anode and platinum cathode. The crystalline precipitate at the anode is Ag<sub>2</sub>CO<sub>3</sub> [P. Demers, Canad. J. Res. (A) <u>17</u>, 77 (1939)].

PROPERTIES:

Light-yellow powder. Very sparingly soluble in water; solubility  $(25 \,^{\circ}C)$ : 3.2  $\cdot 10^{-3}$  g./100 g. H<sub>2</sub>O. Soluble in conc. alkali carbonate solutions, KCN solution, HNO<sub>3</sub> and H<sub>2</sub>SO<sub>4</sub>. Splits off CO<sub>2</sub> on heating (CO<sub>2</sub> pressure at 218°C = 752 mm.). d<sup>2</sup>/<sub>4</sub>° 6.077. Heat of formation (25°C): -120.8 kcal./mole.

REFERENCE:

G. H. G. Jeffrey and A. W. Warnington. Chem. News <u>132</u>, 373 (1939).

## Silver Nitrite

AgNO<sub>2</sub>

 $AgNO_3 + KNO_2 = AgNO_2 + KNO_3$ 169.9 85.1 153.9 101.1

A solution of 5 parts of  $KNO_2$  is added to a solution of 8 parts of  $AgNO_3$ . The resultant pale yellow precipitate is usually

1048

contaminated with some  $Ag_2O$ , which is removed by recrystallization from water at 70°C. On cooling,  $AgNO_2$  crystallizes in the form of hair-thin, almost colorless needles. It is best to work under red light to prevent decomposition.

PROPERTIES:

Colorless to yellow needles. Somewhat soluble in water; the solubility increases markedly with temperature:  $(15^{\circ}C)$  0.28,  $(60^{\circ}C)$  1.38 g./100 g. H<sub>2</sub>O. Soluble in excess nitrites, with formation of complex salts. Blackens in light. Decomposes at 140°C; dissociates into Ag and NO<sub>2</sub> on dry heating. In aqueous solution, gradually decomposes into Ag, AgNO<sub>3</sub> and NO.  $d_4^{26}$  4.453. Crystal structure: orthorhombic. Heat of formation: -11.6 kcal./mole.

REFERENCE:

J. A. A. Ketelaar. Z. Kristallogr. (A) <u>95</u>, 383 (1936).

## Silver Tartrate

#### $Ag_2C_4H_4O_6$

2 AgNO <sub>3</sub>	$+ KNaC_4H_4O_6$	$= Ag_2C_4H_4O_6$	+ KNO <sub>3</sub>	+ NaNO <sub>3</sub>
	(4 H <sub>2</sub> O)			
339.8	282.2	363.8	101.1	85.0

Stoichiometric quantities of  $AgNO_3$  and potassium sodium tartrate (Rochelle salt) are dissolved in water and the solutions combined. On addition of alcohol (in which silver tartrate is insoluble), the product precipitates as a white, cheeselike deposit which is immediately filtered through a suction or Buchner filter. The precipitate is washed with aqueous alcohol until no further  $Ag^+$  ion is detectable. Further purification may be achieved by crystallization from 80°C water, a small quantity of  $Ag_2O$  being formed in the process. The aqueous solution is filtered and alcohol is again added. The precipitate is filtered, washed first with aqueous, then with absolute alcohol or acetone, and dried in vacuum over  $H_2SO_4$ .

Due to the light sensitivity of the compound, it is best to work in the dark.

Alternate method: A hot, moderately conc. solution of Rochelle salt is added to an  $80^{\circ}$ C dilute solution of AgNO<sub>3</sub>; the reaction (addition) is complete when the continually forming precipitate no longer dissolves. The silver tartrate crystallizes on cooling in the form of fine flakes which acquire a white, metallic sheen on washing.

PROPERTIES:

White powder or crystalline flakes, not entirely stable in daylight. Soluble in dilute nitric acid, sparingly soluble in  $H_2O$  (solubility at 25°C:0.20 g./100 g.  $H_2O$ ); insoluble in alcohol, acetone and ether. Decomposes in ammonia and NaOH, yielding Ag<sub>2</sub>O. Evolves CO<sub>2</sub> on dry heating, leaving pyrotartaric acid and Ag as residue.  $d_1^{15}$  3.432.

REFERENCES:

L. Redtenbacher. Liebigs Ann. 38, 132 (1841); H. Sauer. Unpublished.

## o-Phenanthrolinesilver (II) Persulfate

## [Ag phen<sub>2</sub>] S<sub>2</sub>O<sub>8</sub>

 $2 \operatorname{AgNO_3} + 4 \operatorname{C_{12}H_8N_2} + 3 (\operatorname{NH_4})_2 \operatorname{S_2O_8} \\ 339.8 \quad 720.8 \quad 684.6 \\ = 2 \left[\operatorname{Ag}(\operatorname{C_{12}H_8N_2})_2\right] \operatorname{S_2O_8} + 2 (\operatorname{NH_4})_2 \operatorname{SO_4} + 2 \operatorname{NH_4NO_3} \\ 1320.9 \quad 264.3 \quad 160.1 \\ \end{array}$ 

Two equivalents of an aqueous solution of o-phenanthroline are added to a solution of  $AgNO_3$ . A colorless, gelatinous precipitate is formed; it rapidly turns red-brown on addition of a conc. solution of  $(NH_4)_2S_2O_8$ , and settles on standing as fine crystals. The product is suction-filtered, washed with cold water, and dried, first with alcohol and ether and then in a desiccator. The yield is quantitative.

PROPERTIES:

Formula weight 660.44. Chocolate brown, very stable crystalline powder. Insoluble in water. Readily soluble (without decomposition) in cold conc. nitric acid, yielding a dark brown solution from which the perchlorate may be precipitated by addition of an excess of conc. NaClO<sub>4</sub> solution. Forms AgO in alkali hydroxides.

**REFERENCE:** 

W. Hieber. Ber. dtsch. chem. Ges. 61, 2149 (1928).

# Tris-a,a'-dipyridylsilver (II) Perchlorate

[Ag(dipyr)<sub>3</sub>](CIO<sub>4</sub>)<sub>2</sub>

The reaction between silver nitrate and  $\alpha, \alpha'$ -dipyridyl yields bis- $\alpha, \alpha'$ -dipyridylsilver nitrate [Ag(dipyr)<sub>2</sub>]NO<sub>3</sub>, which is converted into bis- $\alpha, \alpha'$ -dipyridylsilver(II) persulfate [Ag(dipyr)<sub>2</sub>]S<sub>2</sub>O<sub>8</sub> with  $K_2S_2O_8$ . Treatment of this compound with nitric acid yields tris- $\alpha, \alpha'$ -dipyridylsilver (II) nitrate  $[Ag(dipyr)_3](NO_3)_2$ ; addition of NaClO<sub>4</sub> to an aqueous solution of this nitrate yields a precipitate of the corresponding perchlorate.

## A) BIS- $\alpha$ , $\alpha'$ -DIPYRIDYLSILVER NITRATE [Ag(dipyr)<sub>2</sub>] NO<sub>3</sub>

A hot solution of 16.9 g. of  $AgNO_3$  in aqueous ethanol is added to a hot solution of 31.2 g. of  $\alpha, \alpha'$ -dipyridyl in ethanol, yielding a precipitate of  $[Ag(dipyr)_2]NO_3$ . Additional product is recovered upon concentration of the mother liquor. The compound is recrystallized from hot dilute ethanol.

PROPERTIES:

Formula weight 482.27. Yellow needles. Decomposes at 155°C. Decomposes slowly in light. Sparingly soluble in cold and hot water and in common organic solvents (except alcohols).

B) BIS- $\alpha$ ,  $\alpha'$ -DIPYRIDYLSILVER (II) PERSULFATE  $[A_g(dipyr)_2]S_2O_8$ 

The yellow needles of  $[Ag(dipyr)_2]NO_3$  are stirred into an excess of cold, saturated aqueous  $K_2S_2O_8$ ; a deep red-brown precipitate of the complex persulfate is produced. The reaction is complete after two hours of constant stirring. The precipitate is washed with cold water.

PROPERTIES:

Formula weight 612.39. Red-brown precipitate. Decomposes at 137°C. Very sparingly soluble in water, insoluble in the common organic solvents. In air, converts to the corresponding hydrogen sulfate.

C) TRIS- $\alpha$ ,  $\alpha'$ -DIPYRIDYLSILVER (II) NITRATE  $[Ag(dipyr)_3](NO_3)_2$ 

Product (B) is triturated in a mortar with cold nitric acid (d 1.4). The excess acid is removed and the residue extracted with warm water. The brown aqueous extract is treated with an excess of aqueous  $NH_4NO_3$  and cooled with ice, thus precipitating small, dark-brown needles of the dinitrate. The precipitate is purified by solution in warm water and reprecipitation with aqueous  $NH_4NO_3$ .

PROPERTIES:

Formula weight 700.47. Dark-brown needles. Decomposes at 176°C. Soluble in water; the aqueous solution decomposes slowly, evolving  $O_2$ . Powerful oxidant.

D) TRIS- $\alpha$ ,  $\alpha'$ -DIPYRIDYLSILVER (II) PERCHLORATE [Ag(dipyr)\_3](ClO<sub>4</sub>)<sub>2</sub>

This compound is precipitated when  $NaClO_4$  is added to an aqueous solution of  $[Ag(dipyr)_3](NO_3)_2$ . The precipitate is washed with warm water.

PROPERTIES:

Formula weight 775.36. Orange-brown crystals. Detonates on heating. Very sparingly soluble in water.

ANALYSIS:

Determination of the Ag (II) in compounds B-D: the complex salts are treated with cold aqueous KI, yielding iodine:  $AgX_2 + 2 KI \rightarrow 2 KX + AgI + \frac{1}{2} I_2$ . The iodine is titrated with Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub> solution.

**REFERENCES**:

G. T. Morgan and F. H. Burstall. J. Chem. Soc. (London) <u>1930</u>, 2594; H. Kainer. Thesis, Heidelberg, 1952.

## Very Pure Gold

Gold (20 g.) is dissolved in aqua regia in an 800-ml. beaker, and the solution is concentrated to a thick sirup. The nitric acid is expelled by evaporating the solution five times on a steam bath, each time with 20 ml. of hydrochloric acid (4:1). The residue is taken up in 650 ml. of hot water and digested until all soluble material is dissolved. It is then allowed to settle for eight days in a dust-free atmosphere. The precipitate consists of AgCl containing small amounts of Au, Pd, SiO<sub>2</sub>, etc. The gold solution is filtered through a double layer of thick filter paper, without disturbing the precipitate. This and all later precipitates are not worked up further to obtain gold.

Experience indicates that the gold refined by use of  $SO_2$  still contains some Pd, while that precipitated with oxalic acid contains Cu, Pb and other metals. Therefore both of these procedures must be used to obtain gold of the desired purity. Sulfur dioxide is passed through the warm gold solution (80°C) obtained above; the gold precipitates quantitatively on careful neutralization with ammonia (1:1). The product is allowed to settle overnight and the deposit of spongy gold is washed by decantation with hot water; it is then heated for four hours on a steam bath with conc. hydrochloric acid and washed free of acid with hot water. Then it is redissolved

1052

in a beaker. The entire procedure is repeated eight times in order to remove Ag, Cu, Ni, Zn and Pb. The product is then digested for 12 hours with ammonia (1:1), washed free of ammonia with water, heated for four hours on a steam bath with hot conc. nitric acid, and decanted. Ammonia (1:1) is again added and later removed by washing with water. The gold sponge is dissolved in dilute aqua regia; after addition of HCl, the solution is concentrated by evaporation, diluted with H<sub>2</sub>O, decanted and filtered. The gold is precipitated by careful addition (there is a danger of foaming over) of small portions of powdered, crystalline oxalic acid. If the solution retains a yellow color, it is carefully neutralized with ammonia and more oxalic acid is added; the addition of the acid is continued until the solution remains colorless. The resultant gold sponge is dissolved and reprecipitated with oxalic acid. It is then Pd-free. Finally the gold is redissolved, precipitated with SO<sub>2</sub>, digested with conc. hydrochloric acid, and washed with water. The last traces of HCl are removed with ammonia. The product is transferred to a glazed porcelain dish and dried at 110°C. Yield 75-80%. The gold prepared in this manner is spectroscopically pure (free of metallic Cu, Ag, Ni, Zn and Pt).

Alternate methods: a) Preparation of pure gold by the method of G. Krüss. The product is probably not as pure as that prepared by the method described above [G. Krüss, Liebigs Ann. 238, 43 (1887)].

b) Extraction of AuCl<sub>3</sub> with ether. Total impurities in the product, about 0.001% [F. Mylius, Z. anorg. allg. Chem. <u>70</u>, 203 (1911)].

PROPERTIES:

Atomic weight 197.0. M.p. 1063°C, b.p. 2960°C;  $d_4^{17.5}$  19.29. Crystal structure: type A1.

REFERENCE:

T. A. Wright. Metals and Alloys 3, 146 (1932).

## Colloidal Gold

A mixture of 120 ml. of very pure, twice-distilled water (silver condenser) and 2.5 ml. of a solution of 6 g. of  $HAuCl_4 \cdot 4 H_2O$  in one liter of water is prepared and brought to a boil as rapidly as possible. Shortly before the boiling point is reached, 3 ml. of 0.18N solution of very pure potassium carbonate is added. As soon as the solution begins to boil, it is rapidly swirled around

(or stirred with a Pyrex glass rod), while 3-5 ml. of a dilute solution of formaldehyde (1.3 ml. of 40% formaldehyde in 100 ml. of water) is added. Stirring is continued until a reaction is evident (this usually occurs within a few seconds—one minute at most), whereupon the solution becomes bright red. It is again brought to a boil and held at the b.p. for a short time until the odor of formaldehyde disappears.

Alternate methods: a) Reduction with ethereal solution of phosphorus (R. Zsigmondy and P. A. Thiessen, Das kolloide Gold [Colloidal Gold], Leipzig, 1925, p. 487).

b) Reduction with hydrazine hydrate or phenylhydrazonium chloride [A. Gutbier and F. Resenscheck, Z. anorg. allg. Chem. 39, 112 (1904)].

c) Reduction with sodium anhydro methylene citrate (citramin) [L. Vanino, Kolloid-Z. <u>20</u>, 122 (1917)].

d) Reduction with alkaline formaldehyde solution [P. P. von Weimarn, Kolloid-Z. 33, 75 (1923)].

e) Sol of gold by pulverization in an electric arc [Th. Svedberg, Ber. dtsch. chem. Ges. <u>39</u>, 1705 (1906); G. Bredig, Angew. Chem. 950 (1898); E. F. Burton, Phil. Mag. <u>11</u>, 425 (1906)].

PROPERTIES:

Bright red sol, particle size about  $1-6 \cdot 10^{-7}$  cm. Highly sensitive to electrolytes. Concentrated solutions (up to 0.12% gold) may be obtained by dialysis.

REFERENCE:

R. Zsigmody and P. A. Thiessen. Das kolloide Gold [Colloidal Gold], Leipzig, 1925, p. 33.

## Gold from Residues

I. FROM PLATING BATHS

A clay cell filled with NaCl solution and provided with a zinc electrode is placed in the gold solution. An electrode made of a piece of brass sheet is immersed in the gold solution and the two electrodes are connected into a circuit. The gold is deposited quantitatively on the brass electrode over a period of a few weeks, during which the brass electrode is replaced once or twice and the Zn electrode is pickled several times.

## II. FROM WASTE CONTAINING GOLD AND SILVER

The particles are calcined and the resultant powder is boiled with conc. nitric acid to remove Ag and other metals. The diluted solution is filtered and the residue is heated with aqua regia on a steam bath for 24 hours. The gold is precipitated from the filtrate with  $FeSO_4$  and worked up.

III. FROM INDUSTRIAL ALLOYS

The gold alloy is ground as finely as possible and heated with conc. hydrochloric acid on a sand bath, conc. nitric acid being added in drops from time to time. When solution is complete, the mixture is evaporated in a porcelain dish placed on a steam bath (dust-free atmosphere) until the liquid solidifies on cooling. The residue is taken up in a large quantity of water and allowed to stand for some time; the precipitated AgCl is then filtered off. The solution is heated and the gold is precipitated with excess aqueous FeCl<sub>2</sub>. The supernatant is decanted and the residue boiled with dilute hydrochloric acid until the HCl ceases to yellow. The solution is then filtered, dried and fused with borax in a porcelain crucible. If higher purity is desired, the procedure is repeated. If present, Pt, Pd and Tl may be removed from the filtrate with Fe or Zn.

Alternate methods: a) Reduction with alkaline  $H_2O_2$  [L. Vanino and L. Seemann, Ber. dtsch. chem. Ges. <u>32</u>, 1968 (1899)].

b) The gold solution is added at  $100^{\circ}$ C to a solution of Hg<sub>2</sub> (NO<sub>3</sub>)<sub>2</sub>, yielding very finely divided gold (L. Vanino, Handbuch der präp. Chemie [Handbook of Prep. Chemistry], Stuttgart, 1921, Vol. I, p. 520).

c) Electrolysis of Ag- and Pt-containing alloys [W. Möbius, Berg- und hüttenm. Ztg. <u>44</u>, 447 (1885); <u>47</u>, 324 (1888); Chemiker-Ztg. <u>15</u>, Rep. 18 (1891); E. Wohlwill, Z. Elektrochem. <u>4</u>, 379 (1897)].

**REFERENCES:** 

- I. Plage. Industr.-Bl. 190 (1878).
- II. W. Adolphi. Chemiker-Ztg. 52, 109 (1928).
- III. A. Bender. Anleitung z. Darstellung anorg. Praparate [Introduction to the Preparation of Inorganic Compounds], Stuttgart, 1893.

## Gold (I) Chloride

## AuCl

Prepared by thermal decomposition of an auric chloride obtained from hydrogen tetrachloroaurate (III).

 $\begin{array}{rll} HAuCl_4 &= AuCl + HCl + Cl_2 \\ (4 H_2O) \\ 411.9 & 232.5 & 36.5 & 70.9 \end{array}$ 

Gold (5-10 g.) is dissolved in aqua regia and the solvents are removed by vacuum distillation (aspirator) at water bath temperature.

The solution is protected by a blanket of CO<sub>2</sub> introduced through a capillary. The nitric acid is expelled by double evaporation with conc. hydrochloric acid, and the resultant dark red-brown melt is poured into a dish where it congeals to a crystalline mass. This is heated to 100°C in high vacuum, until no vapor pressure is evident. Since the HAuCl<sub>4</sub> liquefies again during this operation, care should be exercised to avoid spattering. After all water is expelled, the residue is heated to 156°C (boiling bromobenzene bath). At higher temperatures (170-205°C) the decomposition is complete within a few hours.

Alternate methods: a) Thermal decomposition of AuCl<sub>3</sub> in air at 185°C [J. Thomsen, J. prakt. Chem. <u>13</u>, 337 (1876)]. b) Decomposition of AuCl<sub>3</sub> in a stream of dry HCl at 175°C [M. E. Diemer, J. Amer. Chem. Soc. <u>35</u>, 552 (1913)].

c) Decomposition of AuCl, in a stream of dry air [F. H. Campbell, Chem. News 96, 17 (1907)].

The products prepared by methods a-c are not completely pure.

SYNONYMS:

Aurous chloride, gold monochloride.

PROPERTIES:

Light yellow crystals, not deliquescent. M.p. (dec.)  $289^{\circ}$ C;  $d_4^{25}$  7.4. Soluble in alkali chloride solutions. Decomposes on solution in water. Heat of formation  $(25^{\circ}C)$ : -8.4 kcal./mole.

REFERENCE:

W. Biltz and W. Wein. Z. anorg. allg. Chem. 148, 192 (1925).

## Gold (III) Chloride

## AuCI<sub>3</sub>

I.

$$2 \operatorname{Au} + 3 \operatorname{Cl}_{2} = 2 \operatorname{AuCl}_{3}$$
  
394.0 67.2*l*. 606.7

Finely divided gold is treated at 225-250°C (but not higher) with gaseous  $Cl_2$  at a pressure of 900-950 mm. (The gold powder is obtained by precipitating a solution of a gold salt with sulfurous acid, washing and drying at 180°C.) The powder is in a porcelain boat which is placed in a glass tube. At the point immediately adjoining the boat, the tube widens into a sphere with outlets at

top and bottom. Excess  $Cl_2$  escapes through the upper outlet; this outlet also carries a rod, which can be used to push the  $AuCl_3$  (which condenses in the sphere) into a storage flask via the lower outlet. The yield is 0.1-0.2 g. per hour of large (up to 10 mm. long) crystals.

II. About 0.2-0.6 g. of freshly precipitated gold is placed in a 50-ml. flask connected to the atmosphere via a reflux condenser and a drying tower. Molten iodine monochloride is added in drops through a side tube. The reaction starts when the flask is heated. When the reaction subsides, an excess of ICl is added and the mixture is heated for a short time until boiling just begins. The solution is cooled and extracted several times with  $CCl_4$  (distilled over  $P_2O_5$ ). The suspension is then filtered (suction) in a stream of  $N_2$  through a sintered glass funnel; the residue is washed with  $CCl_4$  and freed of  $CCl_4$  in vacuum. The yield is quantitative.

Alternate method:  $HAuCl_4 \cdot 4 H_2O$  is carefully heated in a stream of  $Cl_2$ ; final heating for half an hour at 200°C [M. E. Diemer, J. Amer. Chem. Soc. 35, 553 (1913)].

SYNONYMS:

Auric chloride, gold trichloride.

REFERENCES:

Formula weight 303.37. Ruby-red crystals (when sublimed) or red-brown to dark ruby-red crystalline mass. M.p. 229°C, b.p. (dec.) 254°C;  $d_4^2$ ° 3.9. Melts at 288°C under a Cl<sub>2</sub> pressure of 2 atm. Sublimes at 180°C. Hygroscopic; soluble in H<sub>2</sub>O with formation of H<sub>2</sub>[AuCl<sub>3</sub>O]. The neutral aqueous solution decomposes gradually with separation of metallic gold; acidic solutions are more stable. Soluble in alcohol and ether. Heat of formation (25°C): -28.3 kcal./mole.

**REFERENCES:** 

- M. Petit. Bull. Soc. Chim. France, Mém. <u>37/38</u>, 1141 (1925);
  W. Fischer and W. Biltz. Z. anorg. allg. Chem. <u>176</u>, 81 (1928).
- II. V. Gutmann. Z. anorg. allg. Chem. <u>264</u>, 169 (1951).

## Hydrogen Tetrachloroaurate (III)

## $HAuCl_4 \cdot 4 H_2O$

Precipitated gold is dissolved in aqua regia and the solvent is evaporated at steam bath temperature (aspirated vapor). The nitric acid is expelled by repeating the procedure twice with conc. hydrochloric acid, which is itself removed by evaporation. The resultant melt is poured into a dish, where it congeals to a crystalline mass. The residual mother liquor is decanted, and the crystals are crushed to allow rapid drying in a drying closet. The mass is pulverized several times during the drying operation until it is completely dry.

PROPERTIES:

Formula weight 411.90. Elongated, light-yellow needles, deliquescent in moist air. Soluble in water, alcohol and ether. The anhydrous compound crystallizes from alcohol. One molecule of  $H_2O$  is given off on prolonged standing in dry air. Crystal structure: monoclinic. Heat of formation: -4.5 kcal./mole.

SYNONYMS:

Gold trichloride acid; chloroauric acid; aurochlorohydric acid; hydrochloroauric acid.

**REFERENCES:** 

W. Biltz and W. Wein. Z. anorg. allg. Chem. <u>148</u>, 192 (1925);
 J. Thomsen. Ber. dtsch. chem. Ges. <u>16</u>, 1585 (1883).

Potassium Tetrachloroaurate (III)

 $KAuCI_4 \cdot 1/_2 H_2O$ 

An aqueous solution of  $AuCl_3$  or  $HAuCl_4$ , strongly acidified with HCl, is treated with an equimolar quantity of conc. aqueous KCl, and the mixture is evaporated over  $H_2SO_4$  or at a moderately high temperature.

PROPERTIES:

Formula weight 386.94. Light-yellow needles, stable in air. Soluble in water and alcohol, insoluble in ether. Loses water of crystallization at 100°C. Crystal structure: monoclinic.

SYNONYM:

Auric potassium chloride.

REFERENCE:

H. Topsöe. Ber. Wien. Akad. II, <u>69</u>, 261 (1874).

Gold (III) Oxide

 $Au_2O_3$ 

 $2 \operatorname{Au}(OH)_{3} = \operatorname{Au}_{2}O_{3} + 3 \operatorname{H}_{2}O_{496.0}$ 496.0 442.0 54.0

Gold hydroxide is made according to the procedure outlined in the next preparation and heated to constant weight at 140-150°C.

It is best to start from Au metal if the product must be entirely free of nitrogen oxides. The gold is dissolved in aqua regia and the nitric acid is removed completely by evaporation with hydrochloric acid, repeated five times. The hydroxide is precipitated with a small excess of Na<sub>2</sub>CO<sub>3</sub> (very slight blue color on litmus paper), washed several times with water, centrifuged and purified by electrodialysis for 14 days. The product is dried and converted to Au<sub>2</sub>O<sub>3</sub> at 140-150°C.

Crystalline  $Au_2O_3$  cannot be obtained by dehydration of  $Au(OH)_3$ . Alternate methods: a) By atomization of gold by means of a high-frequency spark in ozonized  $O_2$ . The oxidation product contains about 40%  $Au_2O_3$ , the rest being elemental Au (see Thiessen and Schütza, as well as Schütza and Schütza, in references below).

b) Precipitation of  $Au(OH)_3$  from  $AuCl_3$  solution with potassium hydroxide by Fremy's method [W. E. Roseveare and T. F. Buehrer, J. Amer. Chem. Soc. <u>49</u>, 1221 (1927)].

#### DETERMINATION OF ACTIVE OXYGEN

The solution is boiled in 0.1N oxalic acid and back-titrated with  $KMnO_4$  solution.

SYNONYMS:

Gold trioxide; gold sesquioxide; auric oxide.

PROPERTIES:

Black to brownish black. Soluble in conc. acids, markedly soluble in glacial acetic acid. Decomposes above  $160^{\circ}$ C into Au and O<sub>2</sub>. Heat of formation (25°C): +19.3 kcal./mole.

REFERENCES:

P. A. Thiessen and H. Schütza. Z. anorg. allg. Chem. <u>243</u>, 32 (1939); H. Schütza and I. Schütza. Z. anorg. allg. Chem. <u>245</u>, 59 (1940); G. Lunde. Z. anorg. allg. Chem. <u>163</u>, 345 (1927).

# Gold (III) Hydroxide Au(OH)3

 $2 \text{ KAuCl}_4 + 3 \text{ Na}_2\text{CO}_3 + 3 \text{ H}_2\text{O}$   $(^{1}/_2 \text{ H}_2\text{O})$   $773.9 \quad 318.0 \quad 54.0$   $= 2 \text{ Au}(\text{OH})_3 + 6 \text{ NaCl} + 2 \text{ KCl} + 3 \text{ CO}_2$   $496.0 \quad 350.7 \quad 149.1 \quad 132.0$ 

A solution of KAuCl<sub>4</sub> is heated for several hours on a water bath with an excess of Na<sub>2</sub>CO<sub>3</sub>. The resultant precipitate is filtered, thoroughly washed, digested with warm, dilute sulfuric acid, and carefully washed in a glass filter crucible until the filtrate is free of  $H_2SO_{4*}$ . The product is dried at room temperature over  $H_2SO_{4*}$ .

Alternate methods: a) Precipitation of AuCl<sub>3</sub> solution with MgCO<sub>3</sub> [G. Krüss, Liebigs Ann. 237, 290 (1887)].

b) Hydrolysis of gold sulfate or nitrate [P. Schottländer, Liebigs Ann. 217, 312 (1883)].

c) Fusion of gold with  $Na_2O_2$  and decomposition of the resultant sodium aurate with dilute sulfuric acid [F. Meyer, Compt. Rend. Hebd. Séances Acad. Sci. <u>145</u>, 805 (1907)].

d) Anodic oxidation of gold in 1N sulfuric acid [F. Jirsa and O. Buryànek, Z. Elektrochem. 29, 126 (1923); W. G. Mixter, J. Amer. Chem. Soc. 33, 688 (191)].

SYNONYM:

Auric hydroxide.

PROPERTIES:

Formula weight 248.02. Brown powder. Insoluble in  $H_2O$ , soluble in conc. acids and hot KOH. When dried over  $P_2O_5$  in vacuum, the compound is converted to AuO(OH) (slowly at room temperature, rapidly at 100°C), whereby the color changes, the final one ranging from yellowish red to ocher-brown. Converts to Au<sub>2</sub>O<sub>3</sub> at 140-150°C.

REFERENCE:

R. Lydén. Z. anorg. allg. Chem. 240, 157 (1939).

#### Potassium Aurate

#### KAuO<sub>2</sub> · 3 H<sub>2</sub>O

 $2 \operatorname{Au}(OH)_3 + 2 \operatorname{KOH} + 2 \operatorname{H}_2O = 2 \operatorname{KAuO}_2 \cdot 3 \operatorname{H}_2O$ 496.0 112.2 36.0 644.3

Auric hydroxide is reacted with warm conc. KOH in the absence of atmospheric CO<sub>2</sub>. After filtration the solution is allowed to evaporate in the dark. The precipitated crystals are dried in vacuum over  $H_2SO_4$ .

**PROPERTIES:** 

Formula weight 322.15. Light-yellow needles, soluble in water, giving a highly alkaline reaction. Decomposes on gentle heating, giving off water and oxygen. The residue consists of Au, KOH and  $KO_2$ .

**REFERENCES:** 

F. Meyer. Compt. Rend. Séances Acad. Sci. <u>145</u>, 805 (1907);
 E. Frémy. Ann. Chim. Phys. <u>31</u>, 483 (1853).

#### Gold (I) Sulfide

Au<sub>2</sub>S

 $2 K[Au(CN)_2] + H_2S + 2 HCl = Au_2S + 2 KCl + 4 HCN$ 576.3 22.4*l*. 72.9 426.1 149.1 108.1

A conc. solution of  $K[Au(CN)_2]$ , obtained by treatment of a solution of  $AuCl_3$  with excess KCN, is saturated with  $H_2S$ . Hydrochloric acid is added to the clear solution and the mixture is heated, resulting in the appearance of a brown color. A heavy, rapidly settling precipitate is formed on boiling. This is filtered, washed with water, and then successively with alcohol, ether,  $CS_2$  and finally again with ether. The product is dried to constant weight over  $P_2O_5$ .

The product usually contains sulfur and some moisture, which are difficult to remove. It may be freed of S by dissolving in KCN, filtering and reprecipitating with boiling hydrochloric acid. PROPERTIES:

Brown-black powder when dry, steel-gray when moist. The freshly precipitated compound is readily soluble in water, forming a colloid, particularly in the presence of  $H_2S$ . Easily recoagulated by hydrochloric acid and salts. Insoluble after drying over  $P_2O_5$ . Resistant to conc. hydrochloric acid and  $H_2SO_4$ , as well as to KOH. Oxidized by aqua regia and strong oxidants. Soluble in solutions of KCN and alkali polysulfides. Decomposes at 240°C into Au and S.

REFERENCE:

L. Hoffmann and G. Krüss. Ber. dtsch. chem. Ges. <u>20</u>, 2361 (1887).

## Gold (II) Sulfide

#### AuS

 $8 \operatorname{AuCl}_{3} + 9 \operatorname{H}_{2}S + 4 \operatorname{H}_{2}O = 8 \operatorname{AuS} + 24 \operatorname{HCl} + \operatorname{H}_{2}SO_{4}$ <sup>1</sup>/<sub>10</sub> 242.7 20.2 l. 7.2 183.3 87.5 9.8

A neutral 1-3% solution of AuCl<sub>3</sub> is precipitated in the cold (the temperature must not exceed 40°C) with  $H_2S$  or an alkali sulfide. The precipitate is filtered, thoroughly washed with water, and treated with alcohol, anhydrous ether,  $CS_2$  and again with ether. The product is dried at 130°C over  $P_2O_5$ .

Alternate method: A solution of  $AuCl_3$  is added in drops to an aqueous solution of sodium dithiosulfatoaurate (I) (for preparation, see under  $Au_2C_2$ ) (Antony and Lucchesi, see references below).

PROPERTIES:

Formula weight 229.07. Deep black. Insoluble in water and acids; soluble in aqua regia and solutions of potassium cyanide and alkali polysulfides. Resistant to KOH in the cold, decomposes after prolonged boiling, liberating gold. Thermal decomposition begins at 140°C.

**REFERENCES**:

U. Antony and A. Lucchesi. Gazz. Chim. Ital. <u>19</u>, 552 (1889); <u>20</u>, 61 (1890);
 L. Hoffmann and G. Krüss. Ber. dtsch. chem. Ges. <u>20</u>, 2704 (1887).

1062

## Gold (III) Sulfide

#### Au<sub>2</sub>S<sub>3</sub>

 $2 \text{ HAuCl}_4 + 3 \text{ H}_2\text{S} = \text{Au}_2\text{S}_3 + 8 \text{ HCl}$ (4 H<sub>2</sub>O) 823.8 67.2 l. 490.2 291.7

A fast stream of  $H_2S$  is introduced into 1N hydrochloric acid at -2 to  $-4^{\circ}C$ ; simultaneously, a cooled solution of  $HAuCl_4 \cdot 4 H_2O$ is allowed to flow in. The black precipitate is digested with water, washed free of acid, treated with alcohol and ether, extracted with  $CS_2$  in a Soxhlet extractor, washed with ether, and dried in vacuum over  $P_2O_5$ .

Alternate methods: a) A solution of  $AuCl_3$  in absolute ether is saturated with  $H_2S$ . The product is washed with  $CS_2$  and alcohol [K. A. Hofmann and F. Höchtlen, Ber. dtsch. chem. Ges. <u>37</u>, 245 (1904)].

b) Completely dry  $LiAuCl_4 \cdot 2 H_2O$  is treated with  $H_2S$  at  $-10^{\circ}C$ . The product is extracted with alcohol,  $CS_2$ , and again with alcohol and ether, and dried at 70°C in pure  $N_2$  [U. Antony and A. Lucchesi, Gazz. Chim. Ital. <u>19</u>, 552 (1889)].

PROPERTIES:

Deep black. Insoluble in water. Resistant to hydrochloric and sulfuric acids and dilute nitric acid. Vigorous reaction with conc. nitric acid. Soluble in conc. Na<sub>2</sub>S solution, alkali polysulfides and KCN.  $d_4^{20}$  8.754.

REFERENCE:

A. Gutbier and E. Dürrwächter. Z. anorg. allg. Chem. <u>121</u>, 266 (1922).

## Gold (I) Acetylide

#### Au<sub>2</sub>C<sub>2</sub>

Prepared by precipitating a solution of sodium dithiosulfatoaurate (I), Na<sub>3</sub>[Au(S<sub>2</sub>O<sub>3</sub>)<sub>2</sub>]  $\cdot$  2 H<sub>2</sub>O, with acetylene.

A) SODIUM DITHIOSULFATOAURATE (I)

A solution of 3 parts of  $Na_2S_2O_3 \cdot 5 H_2O$  in 50 parts of water is reacted (stirring) with a solution of 1 part of AuCl<sub>3</sub> in 50 parts of water. The gold solution is added in portions, each portion being added only after the red color resulting from the previous addition has disappeared. The compound is precipitated from this solution with 96% alcohol and purified by repeated solution in water and reprecipitation with alcohol.

B) GOLD (I) ACETYLIDE

A solution of  $Na_3[Au(S_2O_3)_2]$  is reacted with an excess of strong aqueous ammonia and then slowly saturated with  $C_2H_2$ . The solution becomes yellow, and a yellow precipitate deposits after some time. It is washed by decantation with water and alcohol, filtered and dried over  $H_2SO_4$ .

PROPERTIES:

Formula weight 418.02. Yellow powder. Insoluble in water; decomposes in boiling water without evolution of  $C_2H_2$ ; decomposes slightly in hydrochloric acid with evolution of  $C_2H_2$ . Extremely explosive when dry. Detonates at 83°C if rapidly heated.

**REFERENCE**:

A. Mathews and L. L. Watters. J. Amer. Chem. Soc. <u>22</u>, 108 (1900).

## Gold (I) Cyanide

#### AuCN

 $\begin{array}{rll} K[Au(CN)_2] &+ & HCl &= & AuCN &+ & HCN &+ & KCl \\ 288.1 & & 36.5 & & 223.0 & & 27.0 & & 74.6 \end{array}$ 

An aqueous solution of  $K[Au(CN)_2]$  is mixed in the cold with hydrochloric acid and warmed to 50 °C. Most of the AuCN precipitates. The mixture is evaporated to dryness on a steam bath, resulting in removal of HCN. The residue is taken up in water, filtered, thoroughly washed (in the absence of sunlight) to remove KCl, and dried over  $H_2SO_4$  or  $P_2O_5$ .

Alternate methods: a) Precipitation of a solution of AuCl<sub>3</sub> with KCN [P. O. Figuier, J. Pharm. Chim. <u>22</u>, 329 (1836)].

b) Decomposition of Na [Au(CN)<sub>2</sub>] with HCl [A. Wogrinz, Metalloberfläche <u>8</u>, 11, 162 (1954)].

PROPERTIES:

Lemon-yellow crystalline powder. Stable in air. Sparingly soluble in water and dilute acids. Soluble in solutions of alkali

1064

cyanides, KOH, ammonia,  $Na_2S_2O_3$  and  $(NH_4)_2S$ . Decomposes with separation of Au on dry heating. Unstable in light when moist.  $d_4^{20}$  7.12. Crystal structure: hexagonal.

REFERENCE:

K. Himly. Liebigs Ann. 42, 157 (1842).

## Potassium Dicyanoaurate (I)

## K[Au(CN)<sub>2</sub>]

Formed when "fulminating gold" is dissolved.

Pure gold (10 g.) is dissolved in 50 ml. of aqua regia (34 ml. of conc. hydrochloric acid and 16 ml. of nitric acid, d 1.33) on a steam bath. When solution is complete (after about two hours), "fulminating gold" is precipitated by addition of an excess of ammonia. It is washed until Cl-free, dissolved while still moist in a slight excess of KCN solution, concentrated on a steam bath, and allowed to crystallize overnight. Additional salt may be recovered from the mother liquor. The product is recrystallized from an equal amount of boiling water, and dried over  $P_2O_5$  or conc.  $H_2SO_4$ .

Alternate methods: a) Electrolytic solution of Au in warm aqueous KCN [J. Glassford and J. Napier, Phil. Mag. 25, 61 (1844)].

b) To prepare solutions of  $K[Au(CN)_2]$  or  $Na[Au(CN)_2]$  for use in gold-plating baths and still avoid using gold sponge or evolution of HCN, HAuCl<sub>4</sub>, in an amount corresponding to 3 parts by weight of Au, is dissolved in 50 parts by weight of water. The flask contents are swirled around while  $Na_2CO_3$  or  $K_2CO_3$  is added until a test with Congo paper no longer yields a blue color. The gold solution is then poured into a porcelain dish and allowed to react (stirring) with 5.2 parts by weight of NaCN or 6.8 parts of KCN; the solution becomes warm and colorless. Six parts of annealed 0.02-mm.-thick gold foil, cut into small chips, are added, and the mixture is heated for several hours on a water bath with stirring and replenishing of the evaporating water. Residual unreacted gold is removed and the solution is evaporated to dryness [A. Wogrinz, Prakt. Chem. (Vienna) 3, 216 (1952)].

SYNONYM:

Gold potassium cyanide.

PROPERTIES:

Formula weight 288.14. Colorless crystals. Readily soluble in  $H_2O$ , sparingly soluble in alcohol, insoluble in ether and acetone.

Precipitated from saturated aqueous solution by sulfuric acid, hydrochloric acid, nitric acid and alcohol. Decomposes on boiling with acids. Stable in air and light.  $d_4^{2\circ}$  3.45.

REFERENCE:

F. Chemnetius. Chemiker-Ztg. <u>51</u>, 823 (1927).

# SECTION 20

# Zinc, Cadmium, Mercury

## F. WAGENKNECHT AND R. JUZA\*

## Zinc

#### Zn

#### VERY PURE ZINC

Certain grades of commercial zinc are quite pure. The highest purity may be achieved by distillation (Procedure I) or, starting from  $ZnSO_4$ , by purification of the salt and electrolytic isolation of the metal (Procedures I and II). Extreme purity of zinc salts is particularly important in the preparation of scintillators.

#### I. PURIFICATION OF ZINC SULFATE IN SOLUTION

Alumina, standardized by the method of Brockmann, is introduced in portions into a glass tube (30 cm. long, 4 cm. diameter) provided with a fritted glass disk at one end. After each addition the adsorbent is compacted with a glass pestle or by applying a vacuum. The material is allowed to fill 2/3 to 4/5 of the tube length. The material is prevented from fluidizing by placing a piece of filter paper on top. The flask containing the solution is above the column; the liquid flows into the column through an inlet tube bent at an acute angle. The bottom end of the column is placed in a suction flask. Continuous operation of the system is achieved by applying a slight vacuum. The adsorbent removes As, Sb, Bi, Cr, Fe, Hg, U, Pb, Cu and Ag.

To remove Ni or Co, the  $ZnSO_4$  solution is made alkaline with ammonia and filtered through alumina that has been pretreated with an alcoholic solution of, respectively, diacetyldioxime or  $\alpha$ -nitroso- $\beta$ -naphthol.

Manganese may be removed by adding to the  $ZnSO_4$  solution 0.5 g. of  $(NH_4)_2PbCl_6$  hydrolyzed in one liter of redistilled (!)  $H_2O$ , heating the mixture for a short time to the boiling point,

<sup>\*</sup>The second edition was revised by Dr. H. U. Schuster.

and filtering off the deposit of PbO<sub>2</sub> flakes after 24 hours. Traces of the Pb are removed by passing the solution through an  $Al_2O_3$ column and concentrating the product. Solutions purified with  $Al_2O_3$  as above satisfy the most stringent requirements.

#### **II. ELECTROLYTIC SEPARATION**

The electrolyte should contain 40-60 g./liter of Zn (calculated as the sulfate). A piece of silk taffeta serves as the membrane. Pure ZnO is suspended by stirring in the anode chamber. The cathode is Al or Pt, the anode is Pt. The current density is 0.01-0.03 amp./cm.<sup>2</sup> The Zn deposit may be peeled off cleanly from the aluminum sheet by cutting off its edges. Inclusions are removed by fusing with a small quantity of NH<sub>4</sub>Cl in a porcelain crucible. The ingot is pickled clean with HCl and thoroughly washed with distilled water.

## III. DISTILLATION OF METALLIC ZINC

The last step in the purification of the Zn metal is a double vacuum distillation. The operation is carried out in a Vycor tube shaped as in Fig. 280. After the distillation ( $650^{\circ}C$ ), a



## Fig. 280. Distillation of zinc.

slight gray deposit, which is separate from the main body of the distillate, may be observed at b. It contains Cd. Traces of a black, very fluffy impurity are left at a. No impurity deposit is formed at b during the subsequent second distillation. To prevent contamination of the final product

with some heavy, low-volatility components still present, the second distillation is stopped before the material at a is depleted. The resultant Zn product is spectroscopically pure. If a quartz tube is used and larger amounts of Zn (e.g., 30 g.) are distilled, the distillate adheres very strongly to the tube wall.

## PROPERTIES:

Atomic weight 65.38. Bluish white. Solubility in Hg (18°C) 2.2 g. Zn/100 g. Hg. M.p. 419.4°C, b.p. 905.7°C; d 7.133. Hardness 2.5. Crystal structure: type A3 (Mg type). Electrochemical equivalent 1.220 g.  $(amp. \cdot hr.)^{-1}$ .

## REFERENCES:

I. E. Tiede and W. Schikore. Ber. dtsch. chem. Ges. 75, 586 (1942); W. Schikore and E. G. Müller. Z. anorg. Chem. 255,

1068

327 (1948); W. Schikore and T. Pankow. Ibid. 258, 15 (1949).

- II. F. Mylius and O. Fromm. Z. anorg. allg. Chem. 9, 144 (1895).
- III. R. Petermann. Thesis, Bern, 1946; O. Hönigschmid and M. von Mack. Z. anorg. allg. Chem. <u>246</u>, 363 (1941).

## Zinc Hydride

#### ZnH<sub>2</sub>

I.

 $\frac{\text{ZnI}_2 + 2 \text{LiAlH}_4}{319.2} = \frac{\text{ZnH}_2 + 2 \text{AlH}_3 + 2 \text{Li}}{75.8} = \frac{67.4}{67.4} = \frac{59.9}{267.7}$ 

Ether solutions of  $ZnI_2$  and  $LiAlH_4$  (mole ratio 1:2) are mixed at -40 °C or below; a white precipitate forms. The product must be separated immediately by centrifugation to prevent contamination by polymeric (AIH<sub>3</sub>)<sub>x</sub>, which begins to form after some time. The LiI remains in the supernatant solution.

II.  $Zn(CH_3)_2 + 2LiAlH_4 = ZnH_2 + 2LiAlH_3CH_3$ 95.4 75.9 67.4 103.9

Dimethylzinc (0.57 g., ~6 mmoles) is distilled in a completely dry atmosphere into an ice-cold solution of 0.59 g. (~15.6 mmoles) of LiAlH<sub>4</sub> in 10 g. of absolute ether (predried over LiAlH<sub>4</sub>). The white precipitate obtained on heating the mixture to room temperature is filtered, washed several times with absolute ether, and freed of adhering ether by heating to 50 °C in vacuum.

The dimethylzinc needed for this preparation is obtained by heating  $Hg(CH_3)_2$  (p. 1119) with a large excess of zinc shot in a closed tube. At 120°C the yield is quantitative after 64 hours. The dimethylzinc may be distilled at atmospheric pressure in a stream of N<sub>2</sub>. B.p. 46°C.

PROPERTIES:

Solid, white, nonvolatile when pure. Readily oxidized; reacts with  $H_2O$  or humid air, evolving  $H_2$ . This reaction is very vigorous in the case of old preparations, which often ignite spontaneously. Stable for some time at room temperature in dry air; in high vacuum at 90 °C, gradually decomposes to the elements.

REFERENCES:

- I. W. Wiberg, W. Henle and R. Bauer. Z. Naturforschg. <u>6b</u>, 393 (1951).
- II. G. D. Barbaras, A. E. Finholt, H. J. Schlesinger et al. J. Amer. Chem. Soc. <u>73</u>, 4585 (1951).
# Zinc Chloride

## ZnCI<sub>2</sub>

I. Very pure, anhydrous  $ZnCl_2$  is prepared by treating Zn with dry HCl at 700°C in a quartz boat placed in a tube of high-melting glass. At this temperature, the formation and sublimation of ZnCl<sub>2</sub> proceed at sufficiently high rates. The sublimed chloride is collected in a section of the tube which is kept cool for this purpose. Temperatures above 700°C should be avoided, since entrainment of zinc vapor with the chloride may result, a phenomenon recognizable by the appearance of a color in the otherwise colorless sublimate. For additional purification, the chloride may be resublimed in a stream of HCl.

II. The same reaction can be carried out in anhydrous ether. Ether and excess HCl are removed on a steam bath (vacuum).

III.  $Zn + 2 CuCl = ZnCl_2 + 2 Cu$ 65.4 198.0 136.3 127.1

A 6.7% solution of CuCl in pure, dry acetonitrile (distilled several times over  $P_2O_5$ ) is electrolyzed at room temperature with a Pt cathode and a Zn anode (voltage across the terminals is 12 v.). The electrolysis proceeds under a blanket of absolutely dry N<sub>2</sub>. The reaction is complete when a gray coating of Zn is observed on the Cu deposited at the cathode. The solvent is evaporated in vacuo, and the acetonitrile solvate of ZnCl<sub>2</sub> is converted into the unsolvated salt by careful heating. Yield 96-98%.

If the appropriate copper salts are used, the process may also be employed for the preparation of  $ZnBr_2$  and  $ZnI_2$ , and by substituting a cadmium anode for the zinc electrode,  $CdBr_2$  and  $CdI_2$  may be prepared.

Zinc chloride exists in three different crystal modifications. Details on the preparation and the structure of the pure individual modifications are given by H. R. Oswald and H. Jaggi [Helv. Chim. Acta 43, 72 (1960)].

PROPERTIES:

Colorless, highly hygroscopic, small crystals. M.p.  $313^{\circ}$ C, b.p.  $732^{\circ}$ C; d (pycn.) 2.93. Solubility per 100 ml. of H<sub>2</sub>O: (0°C) 208 g., (20°C) 367 g. (d 2.08), (100°C) 614 g. Crystallizes in the

anhydrous form only above 28°C. Soluble in methanol, ethanol, ether, acetone and other organic solvents.

REFERENCES:

- I. O. Hönigschmid and M. von Mack. Z. anorg. allg. Chem. <u>246</u>, 366 (1941); for apparatus, see O. Hönigschmid and F. Wittner. Ibid. <u>226</u>, 295 (1936).
- II. R. T. Hamilton and J. A. V. Butler. J. Chem. Soc. (London) 1932, 2283.
- III. H. Schmidt. Z. anorg. allg. Chem. 271, 305 (1953).

## Zinc Hydroxychloride

## Zn(OH)CI

 $ZnO + ZnCl_2 + H_2O = 2Zn(OH)Cl$ 81.4 136.3 18.0 235.7

Zinc hydroxychloride is one of the basic zinc halides which can be prepared as well-defined crystalline compounds by several methods, for example, by dissolving ZnO in zinc halide solutions of definite concentration.

The compound is prepared by adding 6-7 g. of ZnO to 100 ml. of a 70% solution of ZnCl<sub>2</sub> and heating to about 150°C until solution is complete. (If seeding crystals are present, the solution becomes turbid and crystallization begins at 133°C.) A coarsely crystalline product is obtained by cooling the solution to 50°C (where the first crystals separate), then heating to 135°C and allowing the mixture to cool slowly to room temperature over a period of 24 hours. Most of the crystals deposit between 130 and 100°C. The mass is carefully crushed and washed with acetone until the filtrate exhibits only a weak opalescence on addition of AgNO<sub>3</sub>. The product is dried in vacuum over CaCl<sub>2</sub>.

SYNONYM:

Zinc chloride hydroxide.

PROPERTIES:

Formula weight 117.85. Colorless hexagonal leaflets. The chloride content is removed by water.

REFERENCES:

Driot. Comptes Rendus Hebd. Séances Acad. Sci. <u>150</u>, 1426 (1910); W. Feitknecht. Helv. Chim. Acta <u>13</u>, 22 (1930).

# Ammonium Tetrachlorozincate

## (NH<sub>4</sub>)<sub>2</sub>ZnCl<sub>4</sub>

 $ZnCl_2 + 2 NH_4Cl = (NH_4)_2ZnCl_4$ 136.3 107.0 243.3

A solution of 70 g. of  $ZnCl_2$  and 30 g. of  $NH_4Cl$  is prepared by heating with 29 (!) ml. of hot  $H_2O$ . It is advisable to measure the water with a balance. The homogeneous diammonium salt crystallizes on cooling. Yield 45 g.

The three-component system  $ZnCl_2-NH_4Cl-H_2O$  has been investigated by Meerburg. It was found that  $(NH_4)_2ZnCl_4$  can be precipitated only from solutions which have higher concentrations of  $ZnCl_2$  than the desired salt. A solution containing  $ZnCl_2$  and  $NH_4Cl$  in a 1:2 ratio usually yields the salt  $ZnCl_2 \cdot 3NH_4Cl$ .

PROPERTIES:

Shiny, orthorhombic leaflets, crystallizable only from  $ZnCl_2$  solutions. M.p. ~ 150°C; d 1.88. Crystal class  $D_{2h}$ .

REFERENCE:

P. A. Meerburg. Z. anorg. allg. Chem. 37, 199 (1903).

### Zinc Bromide

#### ZnBr<sub>2</sub>

The purest material is obtained by electrolytically dissolving purified Zn in a mixture of aqueous HBr and  $Br_2$  in a quartz dish. The solution is digested with an excess of Zn, filtered and crystallized by evaporation. The crystals are recrystallized from dilute hydrobromic acid and separated from the mother liquor by centrifugation. The product is then sublimed in a stream of HBr-N<sub>2</sub>.

Alternate method: See under zinc chloride, p. 1070.

**PROPERTIES:** 

Colorless, highly hygroscopic crystals. Sublimes, producing lustrous needles. M.p. 394°C, b.p. 650°C; d 4.201. Solubility (0°C) 388 g. (2  $H_2O$ ); (100°C) 675 g. (anhydrous  $ZnBr_2$ )/100 ml.  $H_2O$ . Anhydrous when crystallized above 37°C. Soluble in alcohol and ether. Crystal structure: tetragonal; space group  $I4_1/acd$ .

**REFERENCES:** 

G. P. Baxter and M. R. Grose. J. Amer. Chem. Soc. <u>38</u>, 868 (1916);
 G. P. Baxter and J. R. Hodges. Ibid. <u>43</u>, 1242 (1921).

## Zinc lodide

ZnI<sub>2</sub>

 $Zn + I_2 = ZnI_2$ 65.4 253.8 319.2

I. One part of zinc dust is digested with three parts of iodine and 10 parts of  $H_2O$  until disappearance of the  $I_2$ . The mixture is filtered and concentrated over  $H_2SO_4$  and NaOH in a vacuum desiccator (N<sub>2</sub> atmosphere). The ZnI<sub>2</sub> which crystallizes out is vacuum-distilled at about 400 °C.

Well-dried ZnI<sub>2</sub>, prepared by the wet method, is sublimed in an oil-pump vacuum. The evolving iodine is expelled from the apparatus. The compound is obtained as a pure white sublimate.

II. One part of zinc dust is refluxed with two to four parts (depending on the quality of the Zn dust) of iodine and 10 parts of absolutely anhydrous ether until the initial coloration of the liquid disappears completely. The residual Zn-ZnO slurry is removed from the ether solution by filtration through a fritted glass filter. Most of the ether is distilled off, leaving a product containing about 0.5 mole of ether per mole; the ether is driven off in vacuum (fanning of the flask with a flame will help).

Alternate method: See under zinc chloride, p. 1070.

PROPERTIES:

Colorless, highly hygroscopic crystals. M.p. 446°C, b.p. 624°C; d 4.736. Solubility (18°C) 432 g.; (100°C) 510 g. (anhydrous salt)/100 ml. H<sub>2</sub>O. Below 0°C,  $ZnI_2 \cdot 2H_2O$  crystallizes out of solution. Soluble in ethanol, ether, acetone and dioxane. Sublimes in vacuum (crystal needles). Decomposes on heating in air. Crystal structure: tetragonal; space group I4<sub>1</sub>/acd.

REFERENCES:

- I. T. J. Webb. J. Phys. Chem. <u>27</u>, 450 (1923); W. Biltz and C. Messerknecht. Z. anorg. allg. Chem. <u>129</u>, 161 (1923).
- II. Unpublished experiments of P. Laurer and R. Platz, Heidelberg.

## Zinc Hydroxide

## (crystalline)

## ε-Zn(OH)2

I.  $ZnO + NaOH + H_2O = NaZn(OH)_3 = Zn(OH)_2 + NaOH$ 81.4 40.0 18.0 139.4 99.4 40.0

Analytical grade ZnO (160 g.) is refluxed in a round-bottom flask containing a solution of 600 g. of NaOH in 300 ml. of  $H_2O$ . After the ZnO is dissolved, the solution is diluted with 300 ml. of  $H_2O$  and cooled to 60 °C. At this point, the volume of the solution is about 900 ml. It is filtered and diluted 10 times with water. Crystalline Zn(OH)<sub>2</sub> separates out after 2-3 weeks. This is filtered, washed first with cold water, then several times with warm water, and dried over conc.  $H_2SO_4$ .

Small needles are formed during the initial stages of crystallization; however, standing converts them into the other crystal form.

II. Amorphous  $Zn(OH)_2$  is prepared by adding the stoichiometric quantity of ammonia to a solution containing a known quantity of  $ZnSO_4$ . The precipitate is filtered and washed thoroughly to remove as much of the adsorbed sulfate as possible [if the  $Zn(OH)_2$ is worked up without preliminary washing, the product consists of basic sulfates]. The moist, washed precipitate is dissolved in the required amount of conc. ammonia. Then  $NH_3$  is slowly separated from the solution by placing the beaker with the ammonia solution together with a beaker with  $H_2SO_4$  under a bell jar. A large quantity of crystals is obtained within a week. It is important that the initial removal of  $NH_3$  be slow; then the resultant crystals are 0.5 cm. long, on the average.

PROPERTIES:

Colorless crystals. In equilibrium with water, stable up to 39°C; decomposes at higher temperature. d 3.053. Crystal structure: type C31 [ $\epsilon$ -Zn(OH)<sub>2</sub> type].

In addition to the stable  $\epsilon$ -Zn(OH)<sub>2</sub>, there are five additional crystalline forms of the compound, which are unstable and convert spontaneously to  $\epsilon$ -Zn(OH)<sub>2</sub>.

**REFERENCES:** 

- I. R. Scholder and G. Hendrich. Z. anorg. allg. Chem. <u>241</u>, 76 (1939).
- II. H. G. Dietrich and J. Johnston. J. Amer. Chem. Soc. <u>49</u>, 1419 (1927).

## Zinc Sulfide

#### ZnS

 $\begin{array}{rcl} ZnSO_4 &+ & H_2S &= & ZnS &+ & H_2SO_4 \\ (7 & H_2O) & & & \\ 287.6 & & 22.4 \ l. & & 97.4 & & 98.1 \end{array}$ 

I. Zinc sulfide is preferably precipitated from a slightly acidic buffered aqueous solution: an aqueous solution of  $ZnSO_4$  is treated with ammonium acetate; it is then saturated with  $H_2S$  with heating and frequent stirring (optimum pH for precipitation: 2-3). The precipitate is allowed to settle and the supernatant is decanted. The precipitate is shaken with 2% acetic acid saturated with  $H_2S$ ; the solid is allowed to settle and the washing is repeated. To obtain an oxide-free product, the filtration and drying should be carried out in the absence of air.

II. Well-crystallized zincblende is obtained from pure, dry precipitated zinc sulfide by heating the sulfide in a stream of nitrogen for eight hours at 600-650 °C. The reactor is a ceramic tube.

Pure wurtzite may be prepared from the same ZnS precipitate by heating in a stream of  $N_2$  for one hour at 1150°C.

PROPERTIES:

White powder. M.p. ~ 1650°C (appreciable volatilization); distills without decomposition at high vacuum (5  $\cdot$  10<sup>-4</sup> mm.). d 4.14. Solubility (18°C) 0.688 mg. (freshly precipitated)/100 ml. H<sub>2</sub>O. Soluble in dilute mineral acids. Hardness 3.5-4 (both modifications). The low-temperature modification (sphalerite) crystallizes in the cubic B3 system (zincblende), the high-temperature modification (wurtzite) in the hexagonal B4 system. Transition point: about 900°C. Grinding at room temperature converts the metastable wurtzite to zincblende. REFERENCE:

H. Platz and P. W. Schenk. Angew. Chem. 49, 822 (1936).

# Zinc Formaldehydesulfoxylate

### Zn(SO<sub>2</sub> · CH<sub>2</sub>OH)<sub>2</sub>

#### $2 ZnS_{2}O_{4} + 4 CH_{2}O + 2 H_{2}O = Zn(SO_{2} \cdot CH_{2}OH)_{2} + Zn(SO_{3} \cdot CH_{2}OH)_{2}$ 387.0 120.1 36.0 255.6 287.6

A 33% solution of  $ZnS_2O_4$  (1300 g.) is added to 600 g. of a 30% formaldehyde solution; the addition is accompanied by a temperature rise to 50°C. The liquid is stirred and the temperature maintained at 60-65°C for some time. The mixture is filtered and set aside for 2-3 days. The clear solution is again filtered and concentrated in vacuum while  $SO_2$  is aspirated in through a boiling capillary. The zinc formaldehydesulfoxylate is the first to precipitate. The crystals are separated from the mother liquor by centrifugation and dried by heating in vacuum.

The trihydrate is obtained at  $60^{\circ}$ C from a solution of 100 g. of the anhydrous salt in 100 ml. of H<sub>2</sub>O, the tetrahydrate by allowing a solution saturated at 20°C to stand for some time.

The  $ZnS_2O_4$  solution required in the preparation cannot be prepared according to the directions given on p. 394, since the latter procedure yields aqueous solutions containing only about 10% ZnS<sub>2</sub>O<sub>4</sub>. In this case it is better to react a mixture of the purest possible Zn dust (200 g.) and H<sub>2</sub>O (400 ml.) with SO<sub>2</sub>, which should be prewashed with an alkaline solution of Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub>. The reaction proceeds according to:

> $Zn + 2SO_2 = ZnS_2O_4$ 65.4 128.1 193.5

and is carried out in a wide-neck Erlenmeyer flask at 35-40 °C (stirring). Initially, the mixture must be cooled; later it should be warmed. After several hours the reaction slurry is allowed to settle and the product is filtered through a Buchner funnel. The concentration of the viscous, unstable solution is determined by titration with a 0.01 M solution of indigo carmine [1 mole of indigo is equivalent to 1 mole of  $S_2O_4^{2-}$ ; for additional analytical methods see G. Panizzon, Melliand Textilber. <u>12</u>, 119 (1931)].

A method for the preparation of secondary zinc formaldehydesulfoxylate is described in M. Bazlen, Ber. Dtsch. chem. Ges. <u>60</u>, 1470 (1927); cf. also K. Jellinek, Das Hydrosulfit [Hydroxysulfite], Part II, Stuttgart, 1912.

#### SYNONYMS:

Primary zinc oxymethanesulfinate; monozinc formaldehydesulfoxylate.

A technical-grade product containing over 90% of the anhydrous compound is available under the names Dekrolin soluble conc. (BASF), water-soluble Hydrosulfit BZ (Ciba), Sulfoxite S conc. (Du Pont), etc.

#### PROPERTIES:

Colorless crystal needles. The anhydrous salt is stable in air. The trihydrate (flakes with a nacreous luster) and the tetrahydrate (rhombohedral leaflets) are more labile. Soluble in  $H_2O$ . The solution acts as a bleaching agent and is quite resistant to acids. The reducing activity increases markedly with temperature; the rH values of a formic acid solution at pH 3 are: (25 °C) 15; (50 °C) 2; (90 °C) a maximum of 0.5. Decomposes on prolonged boiling. The warm solution turns alkaline indanthrene yellow G paper blue and decolorizes an alcoholic solution of neutral red.

**REFERENCES**:

K. Winnacker and E. Weingaertner. Chem. Technologie, Vol. 2, p. 80, Munich, 1950; BIOS Final Report No. 422, London, 1945;
H. von Fehling. Neues Handwörterb. d. Chemie [New Handbook of Chemistry], Vol. X, p. 291, Braunschweig, 1930; A. Schaeffer. Melliand Textilber. <u>30</u>, 111 (1949).

## Ammonium Zinc Sulfate

#### $(NH_4)_2Zn(SO_4)_2 \cdot 6H_2O$

$$(\mathbf{NH}_{4})_{2}\mathbf{SO}_{4} + \mathbf{ZnSO}_{4} \cdot \mathbf{7} \mathbf{H}_{2}\mathbf{O} = (\mathbf{NH}_{4})_{2}\mathbf{Zn}(\mathbf{SO}_{4})_{2} \cdot \mathbf{6} \mathbf{H}_{2}\mathbf{O} + \mathbf{H}_{2}\mathbf{O}$$
  
132.1 287.6 401.7 18.0

A solution of 45.2 g. of  $ZnSO_4 \cdot 7H_2O$  and 20.8 g. of  $(NH_4)_2SO_4$ is prepared in 75 ml. of boiling  $H_2O$ . The solution is filtered through a jacketed funnel heated with hot water. The crystals precipitating from the filtrate are separated from the mother liquor and dried in vacuum over anhydrous ammonium zinc sulfate or  $H_2SO_4$ . Yield 50 g.

PROPERTIES:

Water-clear, efflorescent, monoclinic crystals. Solubility of the anhydrous salt (0°C) 7.3 g.; (20°C) 12.6 g.; (85°C) 46.2 g. per 100 ml. H<sub>2</sub>O. d 1.93. Space group  $C_{2h}^{5}$ .

# Zinc Selenide

#### ZnSe

I. A dilute solution of  $ZnSO_4$ , buffered with ammonium acetate, is added dropwise to a saturated aqueous solution of  $H_2Se$ , while a stream of  $H_2Se$  (from  $Al_2Se_3$  and dilute HCl), diluted with oxygenfree  $N_2$  or  $H_2$ , is passed through the liquid. The precipitation vessel is heated on a steam bath. The excess  $H_2Se$  bubbling out of the solution is absorbed in a wash bottle filled with conc. nitric acid. If the Zn salt solution is introduced too rapidly or in too high a concentration, a white precipitate is formed; it requires a very long time to convert to the yellow ZnSe. Since the yellow ZnSe precipitate is difficult to filter, it is centrifuged and then washed (by centrifugation) first with boiled, weakly ammoniacal  $H_2O$  and then with methanol. It is dried in a vacuum desiccator over CaCl<sub>2</sub>, then at 120°C in a drying pistol over  $P_2O_5$ .

When moist, zinc selenide is very sensitive to air. Therefore, to remove oxidation products the dry product is placed in a tube and heated for 2-4 hours at 600 °C in a stream of  $H_2$  or  $H_2$ Se. A boat containing a small amount of Se is placed ahead of the product. The heating is continued until all the Se in the boat evaporates. The cubic modification is thus obtained.

The hexagonal modification is obtained by treating  $\text{ZnCl}_2$  vapor with  $\text{H}_2\text{Se}$ .

II. ZnSe may be prepared by a dry method from a mixture of 4 g. of ZnO, 2.5 g. of ZnS, and 6 g. of Se according to:

$$2 ZnO + ZnS + 3 Se = 3 ZnSe + SO_2$$
  
162.7 97.4 236.9 433.0 64.1

The mixture is heated for 15 minutes at 800°C in a covered quartz crucible.

It is also possible to start from 5 g. of ZnS and 6.5 g. of  $H_2SeO_3$ , and then proceed as above. The reaction is formulated as:

$$ZnS + SeO_2 = ZnSe + SO_2$$
  
97.4 111.0 144.3 64.1

PROPERTIES:

Lemon-yellow powder. Soluble in fuming hydrochloric acid with evolution of  $H_2$ Se. d (pycn.) 5.30. Crystal structure: type B3 (zincblende type) or B4 (wurtzite type). REFERENCES:

- I. R. Juza, A. Rabenau and G. Pascher. Z. anorg. allg. Chem. 285, 61 (1956); Fonzes-Diacon. Comptes Rendus Hebd. Séances Acad. Sci. 130, 832 (1900).
- II. A. Schleede and J. Glassner. German Patent 699,320 (1938), issued to Telefunken Co.

# Zinc Amide

### Zn(NH<sub>2</sub>)<sub>2</sub>

 $\frac{\text{Zn}(\text{C}_2\text{H}_3)_2}{123.5} + 2 \text{NH}_3 = \frac{\text{Zn}(\text{NH}_2)_2}{34.1} + 2 \text{C}_2\text{H}_6 + 2 \text{C}_2\text{H}_6}{97.4}$ 

The preparation is carried out in the apparatus of Fig. 281. Diethylzinc is introduced into the storage vessel through the side tube, a blanket of CO<sub>2</sub> being provided; the side tube is then sealed. For each run, about 3 g. of  $Zn(C_2H_5)_2$  is vacuum-distilled from a to b. The apparatus is then filled through stopcock c with very pure  $N_{2}$ . The tube connecting the two vessels is broken at d, and 50 ml. of absolute ether, carefully dried with Na wire, is added through e. During these manipulations, the system is flushed with a fast stream of  $N_2$ , which is introduced at c and leaves the apparatus through a CaCl<sub>2</sub> tube attached at d. The Zn(NH<sub>2</sub>)<sub>2</sub> is precipitated from the ether solution by a stream of carefully purified  $NH_3$ . Simultaneously, the ether in b is evaporated, an operation which requires about two hours. The product is comminuted by shaking (glass slug f is already present in b). Ammonia is then passed over the product for five hours at 150°C and for 12 hours at room temperature.



Fig. 281. Preparation of zinc amide. *a* storage vessel for diethylzinc; *f* glass slug.

PROPERTIES:

Colorless, amorphous; decomposes slowly in air. d 2.13.

REFERENCE:

R. Juza, K. Fasold and W. Kuhn. Z. anorg. allg. Chem. <u>234</u>, 86 (1937).

## Zinc Nitride

## $Zn_3N_2$

A porcelain boat containing  $\sim 7$  g. of zinc dust is placed in a Vycor tube. The material is heated in a rapid stream of NH<sub>3</sub> for 17 hours at 500°C, for eight hours at 550°C, and finally for 16 hours at 600°C. In the process, about 3 g. of Zn is lost by distillation. The remainder is converted to Zn<sub>3</sub>N<sub>2</sub>. This procedure assumes that the zinc does not fuse into a solid mass, even though it requires temperatures above the melting point for complete conversion to the nitride.

PROPERTIES:

Gray-black; quite stable in air. d (x-ray) 6.40. Crystal structure: type  $D5_3$  (Mn<sub>2</sub>O<sub>3</sub> type).

**REFERENCE:** 

R. Juza, A Neuber and H. Hahn. Z. anorg. allg. Chem. 239, 273 (1938).

## Zinc Phosphides

#### Zn<sub>3</sub>P<sub>2</sub>, ZnP<sub>2</sub>

3 Zn +	- 2 P	$= Zn_3P_2$	Zn	+ 2P	$= ZnP_2$
196.1	62.0	258.1	65.4	62.0	127.3

I. Weighed quantities of zinc and a very slight excess of red phosphorus (total about 12 g.) are slowly heated to 700 °C in an evacuated quartz tube, about 12 cm. long and 10-12 mm. I.D.,

placed in an electric furnace. One end of the tube is allowed to project from the furnace to condense the volatilized phosphorus, which when liquid reacts very rapidly with the zinc. The  $Zn_3P_2$  is then heated to 850°C, sublimed to the other end of the tube, which is maintained at 760°C, and kept at this temperature for about one day. A dense, homogeneous sublimate is obtained.

II. A mixture of  $Zn_3P_2$  and  $ZnP_2$  is obtained by passing phosphorus vapor over hot zinc (the procedure is outlined in the case of  $Zn_3As_2$ , method I; see p. 1083).

PROPERTIES:

 $Zn_3P_2$ : Gray. d (x-ray) 4.54. Evolves  $PH_3$  with acids. Crystal structure: tetragonal, type  $D5_9$  ( $Zn_3P_2$  type).

 $ZnP_2$ : Orange to red needles. d (x-ray) 3.51. Sublimes without decomposition in an atmosphere containing phosphorus vapor; insoluble in nonoxidizing acids. Crystal structure: tetragonal; space group  $D_4^4$  or  $D_4^8$ .

**REFERENCES**:

I.

I. R. Juza and K. Bär. Z. anorg. allg. Chem. <u>283</u>, 230 (1956).
 II. M. von Stackelberg and R. Paulus. Z. phys. Chem. (B) <u>28</u>, 427 (1935).

Zinc Phosphate

## $Zn_3(PO_4)_2 \cdot 4 H_2O$

 $\begin{array}{rrrr} 3\,\text{ZnSO}_4 \cdot 7\,\text{H}_2\text{O} &+& 2\,\text{Na}_2\text{HPO}_4 \cdot 2\,\text{H}_2\text{O} \\ && 862.7 & & 356.0 \end{array}$ 

 $= Zn_{3}(PO_{4})_{2} \cdot 4 H_{2}O + 2 Na_{2}SO_{4} + H_{2}SO_{4} + 21 H_{2}O$ 458.2 284.1 98.1

A solution of 5.8 g. of  $ZnSO_4 \cdot 7H_2O$  in 400 ml. of  $H_2O$  is stirred at the boiling point with a solution of 2.5 g. of  $Na_2HPO_4 \cdot 2H_2O$ in 100 ml. of  $H_2O$ . The crystalline precipitate which forms immediately is analytically pure.

II.  $3 ZnO + 2 H_3PO_4 + H_2O = Zn_3(PO_4)_2 \cdot 4 H_2O$ 244.1 196.0 18.0 458.2

A 69% solution of  $H_2 PO_4$  (d 1.52, 100 g.) is saturated at the boiling point (121°C) with ZnO (about 42 g.), taking care to replenish the evaporated water. The solution is then cooled to room

temperature and finally placed in ice. Ten parts (by volume) of ice-cold water is added with vigorous stirring and the solution is filtered into a porcelain dish, in which it is heated (with stirring) on a steam bath. The transparent lamellae of the tetrahydrate precipitate after a short time; they are suction-filtered, washed with boiling water, and dried on a clay plate. Yield 16 g.

#### PROPERTIES:

Colorless crystals, needle-shaped and tabular. Solubility in  $H_2O$  decreases with increasing temperature; can be recrystallized only from solutions containing phosphoric acid. Soluble in dilute acids and dilute ammonia. Loses two moles of  $H_2O$  at 100°C, a third mole at 190°C; the anhydrous salt is obtained at about 250°C. d 3.109. Hardness 2-3. Crystal structure: orthorhombic.

SYNONYM:

Zinc orthophosphate.

REFERENCES:

- I. E. Thilo and J. Schulz. Z. anorg. allg. Chem. 265, 201 (1951).
- II. N. E. Eberly, C. V. Gross and W. S. Crowell. J. Amer. Chem. Soc. <u>42</u>, 1432 (1920).

# Zinc Hydroxyphosphate

## Zn<sub>2</sub>(OH)PO<sub>4</sub>

 $Zn_3(PO_4)_2 + ZnO + H_2O = 2 Zn_2(OH)PO_4$   $4 H_2O$ 458.2 81.4 18.0 485.5

An intimate mixture of 1.146 g. (0.0025 mole) of  $Zn_3(PO_4)_2$ . 4H<sub>2</sub>O (cf. p. 1081) and 1.63 g. (0.02 mole) of ZnO is placed in a porcelain crucible and covered with water; the crucible is half full at this point. The crucible is heated in an autoclave for seven hours at 190°C and 12 atm. The product is digested with 8% methanolic acetic acid on a fritted glass filter and is then washed until the filtrate is free of Zn. After drying at 100°C, the product is analytically pure.

PROPERTIES:

Formula weight 242.75. The colorless crystals are identical with the mineral tarbutite. The water of hydration is given off above 450°C. Crystal structure: triclinic.

REFERENCE:

E. Thilo and I. Schulz. Z. anorg. allg. Chem. 265, 201 (1951).

## Zinc Arsenides

#### Zn<sub>3</sub>As<sub>2</sub>, ZnAs<sub>2</sub>

3 Zn	+	2 As	=	Zn <sub>3</sub> As <sub>2</sub>	Zn	-+-	2 As	=	ZnAs <sub>2</sub>
196.1		149.8		346.0	65.4		149.8		215.2

I. A Vycor tube containing a porcelain boat with pure zinc is heated to 700°C in an electrical furnace; the atmosphere in the tube consists of dry, pure  $N_2$  or  $H_2$ . The As, in a second porcelain boat, is placed at the end of the tube which projects out of the furnace and is heated with a gas flame. The As vapor thus produced is carried over the metal by the stream of  $N_2$  or  $H_2$ . Since the metal has already an appreciable vapor pressure at 700°C, crystals of  $Zn_3As_2$  form on the boat rim and on the tube wall, while the unevaporated metal in the boat is converted to a grayblack mass of arsenide.

II. Heating stoichiometric quantities of Zn and As in an evacuated, sealed Vycor bomb at 780°C yields  $Zn_3As_2$ . The same conditions will produce  $ZnAs_2$ , provided an excess of As is used, since the vapor pressure of As in  $ZnAs_2$ , which results in decomposition of the latter, is quite high at the above temperature.

#### PROPERTIES:

 $Zn_3As_2$ : Gray. Gives off  $AsH_3$  with acids. M.p. 1015°C; d (x-ray) 5.62. Sublimes at the m.p. to give needles or lamellae. Possesses metal-type conductivity. Hardness 3. Crystal structure: type  $D5_9$  ( $Zn_3P_2$  type).

ZnAs<sub>2</sub>: Gray black. Orthorhombic crystals. M.p. 771°C. Sublimes at the m.p. Hardness 3. d (x-ray) 5.08.

REFERENCES:

M. von Stackelberg and R. Paulus. Z. phys. Chem. (B) <u>28</u>, 427 (1935);
 W. Heike. Z. anorg. allg. Chem. <u>118</u>, 264 (1921).

## Zinc Thioantimonate

### $Zn_3(SbS_4)_2$

 $3 \text{ZnCl}_2 + 2 \text{Na}_3 \text{SbS}_4 \cdot 9 \text{H}_2 \text{O} = \text{Zn}_3 (\text{SbS}_4)_2 + 6 \text{NaCl} + 18 \text{H}_2 \text{O}$ 408.9 962.3 696.1 350.7 324.3

A solution of 25 g. of Schlippe's salt (see p. 619) in 75 ml. of  $H_2O$  is treated with a solution of 10.6 g. of  $ZnCl_2$  (or 22.5 g. of

 $ZnSO_4 \cdot 7H_2O$ ) in 50 ml. of  $H_2O$ . The chrome yellow precipitate is washed several times by centrifugation with hot water. It is dried at 80°C, then at 100°C; the orange product is ground. It contains about 6% free S, which is extracted in a Soxhlet apparatus with  $CS_{2*}$ .

PROPERTIES:

Orange powder. Decomposed by HCl. Discolors at 160°C; loses S at 200°C, forming  $Sb_2S_3$ . The corresponding Cd salt is orange-red, the mercury (II) salt ocher yellow. d (pycn.) 3.76.

REFERENCE:

F. Kirchhof. Z. anorg. allg. Chem. 112, 67 (1920).

# Diethylzinc

# $Zn(C_2H_5)_2$

Zn	+ C	${}_{2}\mathbf{H}_{5}\mathbf{I}$	=	C₂H₅ZnI	$2 C_2 H_5 ZnI$	=	$Zn(C_2H_5)_2$	+	ZnI₂
65.4	18	56.0		221.4	442.7		123.5		319.2

I. The 500-ml. flask *a* of Fig. 282 is charged with 200 g. of dry  $C_2H_5I$  (prepared by heating  $C_2H_5I$  with Na chips, and siphoning off and distilling the liquid) and 200 g. of zinc dust, previously cleaned with acid and dried at 160-180°C in a



Fig. 282. Preparation of diethylzinc. d mercury seal; e ampoule for product storage.

stream of CO2. Dry zinc turnings are then added until the pile of metal projects above the surface of the liquid. A stream of dry, air-free CO2 or  $N_2$  is introduced through b, expelling the air in the system. When the apparatus is filled with inert gas, the tip of tube c is dipped slightly into the mercury in cylinder dand capillary b is rapidly The flask is flame-sealed. then heated in an 80°C water bath. The temperature of the

bath is gradually raised to 96°C, while the tip of c is lowered into the mercury until it reaches about 20 cm. below the level of the metal. If all necessary precautions to exclude moisture have been taken, the reaction starts after about 1-1.5 hours of refluxing.

The reaction is complete after an additional 1.5-2 hours, when no further droplets of  $C_{2}H_{5}I$  condense in the flask (solid  $C_{2}H_{5}ZnI$ ). The sealed capillary *b* is cut open, *d* is removed and replaced by vessel *e*, and a slow stream of the inert gas is passed through the system. The flask is then tilted so that the condenser points downward, and the  $Zn(C_{2}H_{5})_{2}$  is distilled on an oil bath (about 200°C) into *e*, which is then sealed in the usual way. Yield about 92%.

II. In Dennis's procedure, the starting material is a zinc-copper compound prepared by reducing a mixture of 200 g. of Zn dust and 25 g. of finely powdered CuO for 20 minutes at 400 °C in a stream of H<sub>2</sub>; the product must be used immediately. Sufficient contact area between the metal and the  $C_2H_5I$  is achieved by mixing the finely ground metal with an equal amount of dry sand.

III. Larger quantities of  $Zn(C_2H_5)_2$  may be prepared starting from a zinc alloy containing 5-8% Cu, which is prepared by fusing Zn with brass, casting into rods and cutting into chips. When this alloy is used, one half the necessary quantity of the quite expensive  $C_2H_5I$  may be replaced by  $C_2H_5Br$ . The reaction is then less vigorous.

The product  $Zn(C_2H_5)_2$  is freed of ethane and  $C_2H_5I$  by lowpressure fractional distillation. It is stored in sealed ampoules or in a flask provided with a well-greased stopcock.

The same procedure may be used for the preparation of: di-n-propylzinc, b.p. (9 mm.) 40°C; di-n-butylzinc, b.p. (9 mm.), 81°C; and diisopentylzinc, b.p. (12 mm.) 100-103°C.

### PROPERTIES:

Colorless liquid. M.p. -30 °C, b.p. (760 mm.) 117.6 °C, (30 mm.) 27 °C, (4 mm.) 0 °C; d (20 °C) 1.207, (8 °C) 1.245. Resistant to CO<sub>2</sub>; ignites in air. Decomposes extremely violently in H<sub>2</sub>O, forming Zn(OH)<sub>2</sub> and C<sub>2</sub>H<sub>6</sub>. Soluble in ether.

REFERENCES:

- I. E. Krause and A. von Grosse. Chemie d. metallorgan. Verbindungen [Chemistry of Organometallic Compounds], Berlin, 1937 [preparative directions cited from Simonovich. Zh. Russ. Fiz.-Khim. Obsch. <u>31</u>, 38 (1899)].
- II. L. M. Dennis. Z. anorg. allg. Chem. <u>174</u>, 133 (1928).
- III. Organic Syntheses. Coll. Vol. 2, New York and London 1943/50, p. 184; H. Grubitsch. Anorgan.-präp. Chemie [Preparative Inorganic Chemistry], Springer, Vienna, 1950, p. 458; A. W. Laubengayer and R. H. Fleckenstein. Z. anorg. allg. Chem. 191, 283 (1930).

# Zinc Carbonate

## ZnCO<sub>3</sub>

I.  $ZnSO_4 + 4 KHCO_3 = ZnCO_3 + K_2SO_4 + K_2CO_3 + 2 CO_2 + 2 H_2O$ (7 H\_2O) 287.6 400.5 125.4 174.3 138.2

Neutral  $ZnCO_3$  is obtained when zinc carbonate, precipitated at as low a temperature as possible, is allowed to age for a long time at low, gradually increasing temperature in a  $CO_2$ -free atmosphere.

A IN KHCO<sub>3</sub> solution (300 ml.), cooled to  $3^{\circ}$ C and saturated with CO<sub>2</sub>, is added with stirring to 700 ml. of a 0.1M ZnSO<sub>4</sub> solution at the same temperature. The temperature is maintained below  $10^{\circ}$ C during the first 3-4 days; it is then raised to  $20^{\circ}$ C and maintained there for an additional 2-3 days until the initial flaky precipitate has been transformed into a finely crystalline deposit. The product is washed several times by decantation with water, taking care to remove the flocculent material floating in the supernatant liquor, and washed free of sulfate on a filter. It is dried in a desiccator at room temperature, or by heating at  $130^{\circ}$ C.

The x-ray powder pattern of the product heated at  $130^{\circ}C$  corresponds to that of natural smithsonite (ZnCO<sub>3</sub>), but contains seven additional lines. The yield is satisfactory.

II.  $ZnCl_{2} + 4 KHCO_{3} + x CO_{2}$  136.3 400.5  $= ZnCO_{3} + 2 KCl + K_{2}CO_{3} + (x + 2) CO_{2} + 2 H_{2}O$  125.4 149.1 138.2

Preparation by rapid aging at moderate temperature under  $CO_2$  pressure: 10 ml. of a conc. solution of  $ZnCl_2$  is frozen with Dry Ice in a freezing tube. A fourfold excess of solid KHCO<sub>3</sub> and 10 ml. of H<sub>2</sub>O are added. A few pieces of Dry Ice are added on top and the tube is melt-sealed while still cold. It is kept at room temperature until the contents melt. The tube is then held at 130°C for two hours.

Departures from the above two procedures result in basic products.

PROPERTIES:

Colorless. Converts to the basic salt on boiling with water. Thermal decomposition begins at 140°C; at 295.5°C the pressure of CO<sub>2</sub> is 700 mm. Solubility  $5.7 \cdot 10^{-5}$  g./100 ml. H<sub>2</sub>O. Readily

soluble in acids. d (pycn.) 4.4; d (x-ray) 4.51. Hardness 5 (natural zincspar). Crystal structure: rhombohedral, type  $GO_1$  (calcite type).

REFERENCE:

G. F. Hüttig, A. Zörner and O. Hnevkovsky. Monatsh. Chem. <u>72</u>, 31 (1939).

## Zinc Acetate

### Zn(CH<sub>3</sub>COO)<sub>2</sub>

$\operatorname{Zn}(\operatorname{NO}_3)_2 \cdot 6 \operatorname{H}_2O$	$+ (CH_3CO)_2O =$	$Zn(CH_3COO)_2$	+ 2 HNO <sub>3</sub>	$+ 5 H_2O$
297.5	102.1	183.5	126.0	90.1

A mixture of 10.2 g. of  $Zn(NO_3)_2 \cdot 6H_2O$  and 40 ml. of acetic anhydride is heated. When the vigorous reaction ceases, the mixture is stored in the cold for some time; the crystal slurry is then suction-filtered, washed with some acetic anhydride and ether, and dried in vacuum over KOH and  $H_2SO_4$ . Yield 95%.

PROPERTIES:

Colorless, hexagonal, prismatic crystals. M.p. 242°C. Sublimes in vacuum without decomposition at lower temperatures; decomposes at temperatures higher than the m.p. d 1.84. Sparingly soluble in cold water, dissolves slowly in warm water. The dihydrate crystallizes from dilute acetic acid, the monohydrate from water and absolute alcohol.

REFERENCE:

E. Späth. Monatsh. Chem. 33, 240 (1912).

## Zinc Cyanide

#### Zn(CN)<sub>2</sub>

I.

$$ZnSO_4 + 2 KCN = Zn(CN)_2 + K_2SO_4$$
(7 H<sub>2</sub>O)
287.6 130.2 117.4 174.3

A solution of 10 g. of  $ZnSO_4 \cdot 7H_2O$  in 100 ml. of  $H_2O$  is mixed (constant stirring) with a KCN solution until no further precipitate is formed (about 5 g. of KCN in 50 ml. of  $H_2O$  is needed). The precipitate, which settles well on prolonged boiling, is washed repeatedly with hot  $H_2O$  and dried either with alcohol and ether or at 70°C. Yield about 4 g.

II.  $Zn(CH_3COO)_2 + 2 HCN = Zn(CN)_2 + 2 CH_3COOH$ 183.5 54.1 117.4 120.1

The  $Zn(CN)_2$  is precipitated with hydrocyanic acid from a solution of  $Zn(OH)_2$  in  $CH_3COOH$ . After washing, the product is dried at 110°C.

PROPERTIES:

White, amorphous powder or shiny, rhombic prisms. Insoluble in  $H_2O$  and alcohol. Soluble in alkali cyanides and aqueous ammonia; soluble in dilute acids (evolution of HCN). Decomposes at 800°C; d 1.852. Crystal structure: type C3 (Cu<sub>2</sub>O type).

**REFERENCES**:

- I. Ullmann. Enzyklopädie d. techn. Chem. [Encyclopedia of Ind. Chem.], 2nd ed., <u>10</u>, 718; Loebe. Thesis, Berlin, 1902.
- II. W. Biltz. Z. anorg. allg. Chem. 170, 161 (1928).

## Potassium Tetracyanozincate

K<sub>2</sub>Zn(CN)<sub>4</sub>

I.

 $Zn(CN)_{2} + 2 KCN = K_{2}Zn(CN)_{4}$ 117.4 130.2 247.7

Zinc cyanide is dissolved in the equivalent amount of KCN solution. About 10 min. is required at room temperature; the process may be accelerated by heating. The salt precipitates from the solution on concentrating.

II.  $ZnO + K_2CO_3 + 4 HCN = K_2Zn(CN)_4 + CO_2 + 2 H_2O$ 81.4 138.2 108.1 247.7 44.0

Zinc oxide is suspended in an aqueous solution of the equivalent quantity of  $K_2CO_3$  and treated for several days with gaseous HCN until completely dissolved. Small crystals of the salt complex precipitate from the concentrated filtrate. They are dried at 105 °C.

SYNONYM:

Potassium zinc cyanide, zinc potassium cyanide.

PROPERTIES:

Transparent octahedra. M.p. 538°C; d 1.647. Solubility (20°C) 11 g./100 ml. H<sub>2</sub>O, 1 g./210 g. of 88% v./v. alcohol. Readily soluble in liquid NH<sub>3</sub>. Crystal structure: type B1 (spinel).

**REFERENCES**:

- I. F. Spitzer. Z. Elektrochem. <u>11</u>, 347 (1905).
- II. W. Biltz. Z. anorg. allg. Chem. <u>170</u>, 161 (1928).

## Zinc Silicate

#### Zn<sub>2</sub>SiO<sub>4</sub>

 $2 ZnO + SiO_2 = Zn_2SiO_4$ 162.8 60.1 222.8

I. Two moles of ZnO and one mole of  $SiO_2$  are intimately mixed. The reaction is facilitated by using finely divided starting materials and compressing the mixed powder into 5-g. tablets. The mixture is placed in a platinum boat inside a ceramic protective tube and heated above the melting point of  $Zn_2SiO_4$  (>1512°C) in a Tammann furnace. The protective tube is closed at one end, which helps to exclude the reducing furnace gases to some extent. The reaction may be observed through a port made of cobalt glass. The melting point is reached when the upright raw material tablet collapses. To prevent evaporation of the ZnO, the heating must be rapid.

II. Tablets made of a mixture of two moles of Zno and one mole of amorphous  $SiO_2$  are heated for four days between 900 and 1000 °C. The x-ray powder pattern of the resultant product indicates a homogeneous material.

III. Pneumatolytic-hydrothermal synthesis from ZnO and  $ZiO_2$  in an autoclave at 365 °C.

SYNONYM:

Zinc orthosilicate.

PROPERTIES:

Colorless. Soluble in 20% HF, decomposed by HCl. M.p. 1512°C; d 4.103. Hardness 5.5; crystal structure: type  $S1_3$  (Be<sub>2</sub>SiO<sub>4</sub> type). At 1432°C, forms a eutectic containing one mole of SiO<sub>2</sub>. Phosphoresces on activation with manganese.

**REFERENCES:** 

- I. W. Biltz and A. Lemke. Z. anorg. allg. Chem. 203, 330 (1932).
- II. A. Pabst. Z. phys. Chem. (A) <u>142</u>, 227 (1929).
- III. C. J. van Nieuwenburg and H. B. Blumendahl. Rec. Trav. Chim. Pays-Bas <u>50</u>, 129 (1931).

# Zinc Fluorosilicate

### ZnSiF<sub>6</sub> · 6 H<sub>2</sub>O

 $ZnO + H_2SiF_6 + 5H_2O = ZnSiF_6 \cdot 6H_2O$ 81.4 144.1 90.1 315.5

Somewhat less than the stoichiometric quantity of ZnO is dissolved in aqueous  $H_2SiF_8$ . Complete saturation is avoided because it produces hydrolysis with formation of colloidal silicic acid. The mixture is evaporated on a steam bath in a platinum or lead dish until a film forms on the surface; the film is redissolved with some water and the product is allowed to crystallize over  $H_2SO_4$  in a desiccator.

SYNONYM:

Zinc fluosilicate.

PROPERTIES:

Colorless, rhombohedral prisms, stable in air. Solubility  $(0^{\circ}C)$  50.3 g. of the anhydrous salt,  $(10^{\circ}C)$  52.8 g./100 ml. of saturated solution. A saturated solution at 20°C has d~1.4. d (pycn.) 2.139, d (x-ray) 2.15. Crystal structure: trigonal.

REFERENCES:

W. Stortenbecker. Z. phys. Chem. (A) <u>67</u>, 621 (1909); O. Ruff,
 C. Friedrich and E. Ascher. Angew. Chem. <u>43</u>, 1081 (1930).

Zinc Ferrate (III)

### ZnFe<sub>2</sub>O<sub>4</sub>

I.

$$ZnO + Fe_2O_3 = ZnFe_2O_4$$
  
81.4 159.7 241.1

I. Zinc oxide, precipitated from  $ZnCl_2$  solution and dried in vacuum over  $P_2O_5$ , is mixed with  $\alpha$ - or  $\gamma$ -FeOOH in a ratio of

 $12nO:1Fe_2O_3$ , taking into account the water content. The mixture is then screened and weighed out. The powder is mixed for four hours in a Pyrex bottle on a mechanical shaker. Following this, 4-g. portions of the mixture are placed in an open platinum crucible, which is then set in an electric furnace. The reaction may be carried out either at 800°C in a stream of dry air or at 1000°C in the absence of such an air flow. In either case, one hour is required for the reaction.

When ZnO (prepared by heating  $ZnCO_3$  for two hours at 1000°C) and  $Fe_2O_3$  are used instead of the above-specified raw materials, the mixture must be calcined for six hours at 800°C to obtain a  $ZnFe_2O_4$  with a pure spinel lattice.

II.  $\begin{aligned} \text{ZnCl}_2 &+ 3 \text{ NaOH} &= \text{Na}[\text{Zn}(\text{OH})_3] &+ 2 \text{ NaCl} \\ 136.3 & 120.0 & 139.4 & 116.9 \end{aligned}$  $\begin{aligned} \text{Na}[\text{Zn}(\text{OH})_3] &+ 2 \text{ FeCl}_3 &+ 5 \text{ NaOH} &= \text{ZnFe}_2\text{O}_4 &+ 6 \text{ NaCl} &+ 4 \text{ H}_2\text{O} \\ & & (6 \text{ H}_2\text{O}) \\ 139.4 & 540.6 & 200.0 & 241.1 & 350.7 \end{aligned}$ 

A solution of 2.4 moles of NaOH in 300 ml. of  $H_2O$  is allowed to react with a solution of 0.15 mole of  $ZnCl_2$  in 100 ml. of  $H_2O$ . The resultant Na[Zn(OH)<sub>3</sub>] solution is treated with a solution of 0.3 mole of FeCl<sub>3</sub> · 6 H<sub>2</sub>O and 1.2 moles of HCl in 5000 ml. of H<sub>2</sub>O (vigorous stirring) and, after stirring two hours, heated for 0.5 hour at 60°C. The mixture is allowed to settle and is then allowed to react with 2N NaOH to a permanent red phenolphthalein color. The product is washed by repeated decantation with 2500-ml. portions of H<sub>2</sub>O until the supernatant is free of Cl<sup>-</sup> (about 15 washings are required), filtered through a sintered glass filter, washed until the solid is free of Cl<sup>-</sup>, and dried in a vacuum desiccator over P<sub>2</sub>O<sub>5</sub> and solid KOH. The product then consists of almost black, highly lustrous, brittle pieces. These are crushed, sieved through a 0.15-mm. screen, and redried in the desiccator.

After annealing for one hour at 60 °C, two spinel interferences are barely recognizable in the x-ray powder pattern. The spinel pattern becomes fully developed after heating to 500 °C.

PROPERTIES:

Dry, brown  $ZnFe_2O_4$  is paramagnetic when prepared by either the dry or the wet method. It absorbs more than its equivalent of  $Fe_2O_3$  while maintaining its crystal lattice and becomes ferromagnetic. The magnetizability of these products is maximum at about 70 mole%  $Fe_2O_3$ . d (x-ray) 5.395. Crystal structure: type H1<sub>1</sub> (spinel type). REFERENCES:

R. Fricke and W. Dürr. Z. Elektrochem. <u>45</u>, 254 (1939); G. F. Hüttig, M. Ehrenberg and H. Kittel. Z. anorg. allg. Chem. <u>228</u>, 112 (1936).

## Rinmann's Green

A mixture of the carbonates or oxalates of Zn and Co with an equal amount of KCl (e.g., 15 g. of ZnCO<sub>3</sub>, 3.5 g. of CoCO<sub>3</sub>, and 18.5 g. of KCl) is heated several hours at high temperature (> 1000°C) in a Pt crucible. (The KCl serves as a flux and mineralizer.) The material should then be cooled under a  $CO_2$  blanket. The reaction is brought to completion by repeating the procedure several times followed by washing. The KCl must be replenished between heatings.

At higher temperatures and on vacuum calcination, the color becomes lighter; it is malachite green in the presence of an excess of Zn, brownish pink with an excess of Cn. Products calcined below 1000°C contain green-black  $ZnCn_2O_4$ .

Rinmann's green consists of mixed ZnO-CoO crystals; the green, Co-deficient products (up to about 30% Co) consist of a solid solution of CoO in ZnO (wurtzite lattice). The pink, Co-rich preparations (above 70% CoO) are solutions of ZnO in CoO (NaCl lattice). The intermediate region is heterogeneous.

SYNONYMS:

Cobalt green, turquoise green, cinnabar green.

PROPERTIES:

Soluble in weak acids and solutions of  $(NH_4)_2CO_3$ . d ~ 5.5.

REFERENCES:

J. A. Hedvall. Z. anorg. allg. Chem. <u>86</u>, 201 (1914); C. Natta and L. Passerini. Gazz. Chim. Ital. <u>59</u>, 620 (1929).

Cadmium

(needles)

Cd

$$\begin{array}{rcl} CdSO_4 &=& Cd &+& H_2SO_4 &+& 1/_2O_2 \\ (^{6}/_8 & H_2O) && & \\ 256.5 & & 112.4 & & 98.1 \end{array}$$

Two platinum disk electrodes (diameter 4.5 cm.) are placed one above the other (distance of about 5 cm.) in a vertical glass cylinder (I.D. 7 cm.). The lower electrode serves as the cathode, the upper as the anode. The electrolyte is a conc.  $CdSO_4$  solution slightly acidified with  $H_2SO_4$ . The Cd is deposited as a fine crystal-line powder on the cathode at a current density of 0.1-0.3 amp./cm.<sup>2</sup> The electrolysis vessel fills up quite rapidly with the silvery crystal powder. From time to time the loose powder is compressed with a glass rod to prevent establishment of a short circuit with the anode.

When the Cd in the electrolyte is depleted to such an extent that  $H_2$  begins to evolve at the cathode, the solution must be replenished with CdSO<sub>4</sub> to avoid formation of a spongy deposit (the latter also appears at excessive current densities).

The compound is used as filler in the Jones reductor.

PROPERTIES:

Silvery-white crystal powder. M.p.  $321^{\circ}$ C, b.p.  $765^{\circ}$ C; d 8.642. Bulk density 80%. Solubility (18°C) 5.17 g./100 g. Hg. Soluble in mineral acids. Hardness 2. Electrochemical equivalent 2.097 g.· (amp.-hr.)<sup>-1</sup>. Crystal structure: type A3 (Mg type).

REFERENCES:

F. P. Treadwell. Helv. Chim. Acta <u>4</u>, 551 (1921); F. P. Treadwell. Lehrbuch d. analyt. Chemie [Analytical Chemistry], Vol. 2, Vienna, 1949, p. 542.

# Cadmium Chloride

# CdCI<sub>2</sub>

I.

$Cd(NO_3)_2$	+ 2 HCl	$= \ CdCl_2$	+ 2 HNO <sub>3</sub>
(4 H <sub>2</sub> O) 308.5	72.9	183.3	126.0

Repeated evaporation with very pure conc. hydrochloric acid converts  $Cd(NO_3)_2 \cdot 4H_2O$  to the chloride. The product is recrystallized twice. Partial dehydration is achieved by storing for a prolonged time in a vacuum desiccator containing fused KOH (which is frequently replaced). Final dehydration is achieved by careful heating of the product in a stream of HCl, distilling twice in the same stream, and finally fusing the distillate under pure N<sub>2</sub>.

II.

$$Cd + 2 HCl = CdCl_2 + H_2$$
  
112.4 72.9 183.3 2.0

The reaction between Cd and HCl at 450  $^{\circ}\mathrm{C}$  is smooth and uniform. The chloride is distilled twice in a stream of HCl and melted under N<sub>2</sub>.

A warm solution of about 4 g. of cadmium acetate (dry) in anhydrous acetic acid (or a mixture of the latter with acetic anhydride) is treated with a slight excess of acetyl chlorine or with gaseous HCl. The white precipitate which appears immediately is centrifuged off, washed once or twice with dry benzene, and dried at 100-120°C.

Cadmium bromide may be prepared by the same procedure from cadmium acetate and acetyl bromide (or HBr gas).

### **PROPERTIES:**

Colorless rhombohedral leaflets. M.p. 568°C, b.p. 967°C. Solubility (0°C) 90.1 g. (2.5-hydrate), (20°C) 111.4 g. (2.5-hydrate), (100°C) 150 g. (1-hydrate)/100 ml. H<sub>2</sub>O. Crystallizes as the monohydrate above 34°C. Solubility (15.5°C) 1.7 g. of anhydrous CdCl<sub>2</sub> per 100 g. of ethanol or methanol. d 4.047. Crystal structure: type C 19 (CdCl<sub>2</sub> type). Method III yields a white, microcrystalline powder which in the cold tends to form gelatinous inclusion products with various solvents (e.g., benzene).

REFERENCES:

O. Hönigschmid and R. Schlee. Z. anorg. allg. Chem. 227, 184 (1936); H. D. Hardt. Private communication; A. R. Pray. Inorg. Syn. <u>5</u>, 153 (1957); E. R. Epperson et al. Ibid. <u>7</u>, 163 (1963).

# Cadmium Hydroxychloride

## Cd(OH)CI

Of the five basic cadmium chlorides, Cd(OH)Cl has the highest chloride content; it is the stable end product of the hydrolysis of not too dilute solutions of  $CdCl_2$ .

I.  $CdCl_2 + NaOH = Cd(OH)Cl + NaCl$ 183.3 40.0 164.9 58.5

A 0.1-1M solution of  $CdCl_2$  is treated with 30% of the stoichiometric quantity of aqueous NaOH. The resultant solution should have a pH of 6.6. The precipitate is a labile basic chloride which is converted in stages over a period of a few days to the stable Cd(OH)Cl, provided it is in contact with the mother liquor. The theoretical composition is obtained when a 1M solution of  $CdCl_2$ is used as the starting material.

#### 20. ZINC, CADMIUM, MERCURY

II.  $CdO + CdCl_2 + H_2O = 2 Cd(OH)Cl_{128.4}$  183.3 18.0 329.8

Cadmium oxide is heated for several days at 210 °C with a solution of CdCl<sub>2</sub> in a sealed tube.

SYNONYM:

Cadmium chloride hydroxide.

PROPERTIES:

Colorless, elongated, hexagonal prisms. d 4.57. Layer lattice, type  $E0_3$  [Cd(OH)Cl type].

**REFERENCES:** 

- I. W. Feitknecht and W. Gerber. Helv. Chim. Acta <u>20</u>, 1344 (1937); Z. Kristallogr. (A) <u>98</u>, 168 (1937).
- II. I. L. Hoard and O. D. Grenko. Z. Kristallogr. (A) <u>87</u>, 110 (1934).

# Potassium Cadmium Chloride

#### $CdCl_2 \cdot KCI \cdot H_2O$

 $CdCl_2 + KCl + H_2O = CdCl_2 \cdot KCl \cdot H_2O$ 183.3 74.6 18.0 275.9

This double salt crystallizes below 36.5°C from an aqueous solution of equimolar quantities of the components. The anhydrous salt crystallizes at higher temperature.

The compound is used in the Lipscomb-Hulett standard cell (704 mv.).

PROPERTIES:

Fine silky needles. The saturated solution contains the following amounts of the anhydrous salt:  $(2.6^{\circ}C)$  21.87 g.,  $(19.3^{\circ}C)$  27.50 g.,  $(41.5^{\circ}C)$  35.66 g.,  $(105.1^{\circ}C)$  51.67 g./100 g.

**REFERENCE:** 

H. Hering. Comptes Rendus Hebd. Séances Acad. Sci. <u>194</u>, 1157 (1932).

## Cadmium Bromide

### CdBr<sub>2</sub>

 $Cd + Br_2 = CdBr_2$ 112.4 159.8 272.2

Cadmium is brominated at 450°C in a quartz boat placed inside a Vycor tube initially filled with dry N<sub>2</sub>. Nitrogen is then passed through a washing bottle filled with Br, and introduced into the tube. The complete bromination of 3 g. of Cd requires about two hours at 450°C. Raising the temperature to increase the reaction rate is not recommended, since this may cause appreciable quantities of the metal to distill with the product. The molten CdBr<sub>2</sub> is deep red as long as unreacted metal is present and becomes increasingly lighter as the metal is consumed, so that the end of the reaction may be readily recognized by the final permanent light color. The product CdBr<sub>2</sub> is distilled twice in a stream of Br<sub>2</sub> by raising the temperature; it is freed of excess Br<sub>2</sub> by remelting under pure CO<sub>2</sub>. The entire procedure may be carried out in the apparatus described by O. Hönigschmid and F. Wittner [Z. anorg. allg. Chem. 226, 297 (1936)] for the preparation of pure uranium halides; it is also described under UBr<sub>4</sub> (p. 1440). Alternate method: See under zinc chloride (p. 1070) and cadmium chloride (p. 1093).

PROPERTIES:

Colorless, hexagonal, pearly flakes; highly hygroscopic. M.p. 566°C, b.p. 963°C; d 5.192. Solubility (18°C) 95 g., (100°C) 160 g. per 100 ml. H<sub>2</sub>O. Crystallizes as the monohydrate below 36°C, as the tetrahydrate above this temperature. Solubility (15°C) 26.4 g. of anhydrous  $CdBr_2/100$  g. alcohol. Crystal structure: type C 19 (CdCl<sub>2</sub> type).

REFERENCE:

O. Hönigschmid and R. Schlee. Z. anorg. allg. Chem. 227, 184 (1936).

Cadmium Iodide

## CdI<sub>2</sub>

 $Cd + I_2 = CdI_2$ 112.4 253.8 366.3

Cadmium shavings (or Cd slurry obtained from  $CdSO_4$  solution + Zn) are shaken in distilled  $H_2O$  with the equivalent quantity

I.

of resublimed iodine. The shaking may be dispensed with if the mixture is refluxed for two hours. After the color of the liquid disappears, it is filtered and concentrated on a steam bath. The crystals are vacuum-dried for 24 hours over  $P_{2}O_{5}$  at 100-150°C.

Carefully dried  $CdI_2$  may be sublimed in a stream of oxygenfree  $CO_2$ . The  $CdI_2$  vapor is condensed in a long glass tube closed off with canvas. This yields " $CdI_2$  flowers."

II.

 $\begin{array}{rcl} CdSO_4 &+& 2\,KI &=& CdI_2 &+& K_2SO_4 \\ & & (^{8}\!/_{\!_3} & H_2O) \\ & 256.5 & 332.0 & 366.3 & 174.3 \end{array}$ 

An aqueous solution of three parts of  $CdSO_4 \cdot 8/3 H_2O$  and four parts of KI is evaporated to dryness and extracted with warm absolute alcohol. The  $CdI_2$  crystallizes in colorless lamellae upon cooling of the solution.

Alternate method: See under zinc chloride, p. 1070.

PROPERTIES:

Colorless, lustrous, hexagonal leaflets; stable in air. M.p.  $387 \,^{\circ}$ C, b.p.  $787 \,^{\circ}$ C; d 5.67. Solubility ( $18 \,^{\circ}$ C) 85 g., ( $100 \,^{\circ}$ C) 128 g./100 ml. H<sub>2</sub>O; ( $20 \,^{\circ}$ C) 176 g./100 ml. methanol; ~90 g./100 ml. of ethanol. Soluble in ether. Crystal structure: type C 6 (CdI<sub>2</sub>) and C27 (second CdI<sub>2</sub> type). d (x-ray) of both structures is identical.

**REFERENCES:** 

- I. W. Biltz and C. Mau. Z. anorg. allg. Chem. <u>148</u>, 170 (1925); E. Cohen and A. L. Th. Moesveld. Z. phys. Chem. <u>94</u>, 471 (1920).
- II. Jahresber. Fortschr. d. Chem. 1864, 242.

# Cadmium Hydroxide

#### Cd(OH)<sub>2</sub>

#### I. COARSE CRYSTALS

 $Cdl_2 + 2 KOH = Cd(OH)_2 + 2 KI$ 366.3 112.2 146.4 332.0

A solution of 10 g. of  $CdI_2$  in 200 ml. of water is mixed with 320 g. of carbonate-free KOH. The mixture is heated until the first precipitate of  $Cd(OH)_2$  redissolves at about 135°C. The heating must be accompanied by continuous stirring to prevent

the lower layers of the liquid from reaching a temperature high enough to cause partial conversion of the  $Cd(OH)_2$  to black, sparingly soluble CdO. The major part of the  $Cd(OH)_2$  crystallizes when the solution is slowly cooled. However, a part of the hydrate remains in solution even after complete cooling and may precipitate as amorphous  $Cd(OH)_2$  if the product is immediately treated with water. Therefore, the mixture is allowed to stand for 12 hours before attempting to separate the  $Cd(OH)_2$  with water.

II. Very homogenous  $Cd(OH)_2$  is obtained from cadmium acetate and 85% KOH following precipitation of crystalline CdO by the same procedure.

III. FINE CRYSTALS

A finely crystalline product is obtained by dropwise addition (stirring or shaking) of a boiling solution of  $Cd(NO_3)_2$  to boiling, carbonate-free 0.82N NaOH (stoichiometric quantities). The precipitate is repeatedly washed with hot water and vacuum-dried over  $P_2O_5$  at 60°C. (For details, see in the original.)

PROPERTIES:

Nacreous, hexagonal leaflets soluble in acids and  $NH_4Cl$  solution. Solubility (25°C) 0.26 mg./100 ml.  $H_2O$ ; 0.13 g./100 ml. 5N NaOH. Dehydration starts at 130°C, is complete at 200°C. d 4.79. Crystal structure: type C6 (CdI<sub>2</sub> type).

REFERENCES:

- I. A. de Schulten. Comptes Rendus Hebd. Séances Acad. Sci. <u>101</u>, 72 (1885).
- II. R. Scholder and E. Staufenbiel. Z. anorg. allg. Chem. 247, 271 (1941).
- III. R. Fricke and F. Blaschke. Z. Elektrochem. 46, 46 (1940).

# Cadmium Sulfide

CdS

 $\begin{array}{rcl} CdSO_4 &+ & H_2S &= & CdS &+ & H_2SO_4 \\ & & & \\ \frac{(\theta_1'_8 H_2O)}{256.5} & & 22.4 \ l. & 144.5 & 98.1 \end{array}$ 

Finely divided cubic CdS is obtained by precipitation with  $H_2S$  of a hot,  $H_2SO_4$ -acidified aqueous solution of CdSO<sub>4</sub>. The hexagonal

modification (more or less free of cubic CdS) is obtained from cadmium halide solutions; however, the resultant sulfide is contaminated with strongly adhering halide which cannot be washed out.

Depending on the particle size and the state of the surface, the color of the precipitates varies from lemon yellow to orange. Lemon yellow "cadmium yellow" is prepared by precipitating, with constant stirring, a very diluted neutral solution of  $CdSO_4$ with an excess of Na<sub>2</sub>S solution. The precipitate is then washed free of sulfate.

Dark CdS is obtained by calcining a mixture of two parts of  $CdCO_3$  and one part of sulfur powder in a crucible. The product is pulverized after cooling.

Pure CdS, free of the anions of the precipitating medium, is prepared by bubbling  $H_2S$  through a solution of  $Cd(ClO_4)_2$  in 0.1-0.3N perchloric acid. Lower acid concentrations yield precipitates which are difficult to filter; the precipitation is incomplete at higher concentrations.

Crystals a few millimeters in size are obtained from  $H_2S$  and Cd vapor at about 800 °C (see Frerichs' method in the literature below).

PROPERTIES:

Lemon-yellow to orange powder. Solubility  $(18^{\circ}C) 0.13 \text{ mg.}/100 \text{ ml. }H_2O$ . Soluble in conc. or warm dilute mineral acids. Sublimes at 980°C. d 4.82. Hardness 3. Crystal structure: cubic type B3 (zincblende type) and hexagonal type B4 (wurtzite type). The cubic modification is converted to the hexagonal by heating at 700-800°C in sulfur vapor.

REFERENCES:

W. O. Milligan. J. Phys. Chem. <u>38</u>, 797 (1934); H. B. Weiser and E. J. Durham. Ibid. <u>32</u>, 1061 (1928); W. J. Müller and G. Löffler. Angew. Chem. <u>46</u>, 538 (1933); E. Dönges. Unpublished; G. Denk and F. Denk. Z. analyt. Chem. <u>130</u>, 383 (1949/50); R. Frerichs. Naturwiss. <u>33</u>, 2181 (1946).

CADMIUM SELENIDE CdSe

The preparation is analogous to that of ZnSe (method I, see p. 1078).

PROPERTIES:

Formula weight 191.37. Dark-red powder. d (x-ray) 5.767. Crystal structure: type B3 (zincblende type) and B4 (wurzite type).

## Cadmium Nitride

### $Cd_3N_2$

 $\begin{array}{rcl} 3 \ \mathrm{Cd}(\mathrm{NH}_2)_2 &=& \mathrm{Cd}_3 \mathrm{N}_2 \ + \ 4 \ \mathrm{NH}_3 \\ & 433.4 & 365.3 & 68.1 \end{array}$ 

Cadmium amide,  $Cd(NH_2)_2$ , is thermally decomposed in a vapor pressure eudiometer (see Part I, p. 102) at 180 °C while repeatedly removing measured amounts of NH<sub>3</sub>. The evolution of NH<sub>3</sub> ceases after about 36 hours. The Cd<sub>3</sub>N<sub>2</sub> product decomposes if the temperature is raised higher.

PROPERTIES:

Black; forms oxide in air. d (x-ray) 7.67. Crystal structure: type  $D5_3$  ( $Mn_2O_3$ ).

**REFERENCE**:

H. Hahn and R. Juza. Z. anorg. allg. Chem. 244, 111 (1940).

### Cadmium Amide

### $Cd(NH_2)_2$

 $\begin{array}{rcl} Cd(SCN)_2 \ + \ 2 \ KNH_2 \ = \ Cd(NH_2)_2 \ + \ 2 \ KSCN \\ 228.6 & 110.2 & 144.5 & 194.4 \end{array}$ 

Cadmium thiocyanate (7 g.) is placed on filter disk b of vessel a (Fig. 283). About 15 ml. of carefully purified NH<sub>3</sub> is condensed



Fig. 283. Preparation of cadmium amide. h and k pinch clamps; l storage vessel for ammonia; m glass slug; n glass bulbs. onto the salt, which then dissolves in the ammonia. A solution of  $\text{KNH}_2$  in liquid  $\text{NH}_3$  is added to the above mixture through the ground joint c, and the vessel is closed off with ground cap d. A fluffy, white precipitate of  $\text{Cd}(\text{NH}_2)_2$  is formed. The amount of  $\text{KNH}_2$  used must be somewhat less than stoichiometric as  $\text{Cd}(\text{NH}_2)_2$  dissolves in excess  $\text{KNH}_2$ . In addition, no air must be allowed to be present during the reaction; a stream of  $N_2$  is therefore passed through the apparatus when it is opened for any reason.

After thorough mixing of the two solutions, the supernatant liquid containing KSCN and excess Cd(SCN)<sub>2</sub> in liquid NH<sub>3</sub> is filtered by suction through disk b. This operation is performed by closing stopcock e and carefully evacuating the apparatus through f; this results in transfer of the liquid from a to g, which is cooled with Dry Ice-alcohol. The liquid is then removed from the system and into flask i by application of slight pressure, achieved by closing *f*, raising the temperature in *g* temporarily (remove the cooling bath), and opening the screw pinchcock h. After closing  $h_{i}$ , the product is washed by producing a slight vacuum in the system and transferring fresh liquid ammonia from storage vessel l through screw pinchcock k onto the product in a. Washing is complete when the NH<sub>3</sub> evaporates without leaving a residue. When this point is reached, all the NH<sub>3</sub> is removed (by suction) from the product. The latter is then knocked off the walls by means of the glass slug m and transferred to bulbs n. The preparation is carried out at (or near) the boiling point of NH<sub>3</sub> (-33.5°C). Very pure N<sub>2</sub>, introduced through stopcocks o or f, is used as the blanketing gas.

PROPERTIES:

Slightly yellowish, amorphous; rapidly discolors to brown in air. d 3.05.

REFERENCE:

R. Juza, K. Fasold and W. Kuhn. Z. anorg. allg. Chem. <u>234</u>, 86 (1937).

Cadmium Phosphides

Cd<sub>3</sub>P<sub>2</sub>, CdP<sub>2</sub>, CdP<sub>4</sub>

 $3 \text{ Cd} + 2 \text{ P} = \text{ Cd}_3\text{P}_2$  $337.2 \quad 62.0 \quad 399.2$ 

The procedure for the preparation  $Cd_3P_2$  is similar to that for  $Zn_3P_2$  (method I, see p. 1080); that is, it is produced in an

evacuated quartz tube from Cd metal and red P. The temperature along the tube varies from 400 to 600°C. The mixture is allowed to react for about nine hours, and the entire tube is then heated for 12 hours at 680°C. The resultant  $Cd_3P_2$  is resublimed several times over an open flame, and finally sublimed at 680°C into a slightly colder part of the quartz tube.

II. 
$$Cd + 2P = CdP_2$$
  
112.4 62.0 174.4

A mixture of  $CdP_2$  with  $Cd_3P_2$  is formed by using the procedure given (see p. 1083) for  $Zn_3As_2$  (method I).

III. 
$$Cd + 4P = CdP_4$$
  
112.4 124.0 236.4

A mixture of 0.6 g. of white phosphorus and 20 g. of a Pb-Cd alloy containing 5% Cd is sealed under a  $CO_2$  blanket in a Vycor ampoule. The ampoule is heated in an electric furnace to 565-575°C (the heatup time is a few hours) and maintained at this temperature for 2.5-5 days. If large crystals are desired, the temperature gradient in the furnace should be small and the cooling slow. The CdP<sub>4</sub> is purified by boiling with glacial acetic acid and  $H_2O_2$  and subsequent treatment with 20% hydrochloric acid. The starting Pb-Cd alloy is prepared by fusing the two metals

The starting Pb-Cd alloy is prepared by fusing the two metals under KCN in a porcelain crucible and cutting the product into strips. Commercial phosphorus is purified by melting under dilute chromosulfuric acid and dried under  $CO_2$ .

PROPERTIES:

 $Cd_3P_2$ : Gray, lustrous needles or leaflets. M.p. 700°C; d (x-ray) 5.60. Soluble in hydrochloric acid with evolution of PH<sub>2</sub>, explosive reaction with conc. nitric acid. Crystal structure: tetragonal, type  $D5_9$  (Zn<sub>3</sub>P<sub>2</sub>).

 $CdP_2$ : Orange to red [appears occasionally in an indigo blue modification: B. Renault, Comptes Rendus Hebd. Séances Acad. Sci. <u>76</u>, 283 (1873)]. Tetragonal needles. d (x-ray) 4.19.

 $CdP_4$ : Black, highly reflecting crystals. Very unreactive, dissolves in boiling aqua regia. Decomposes into the elements on heating in vacuum. d (pycn.) 3.90. Crystal structure: monoclinic, space group  $C_{2h}^5$ .

REFERENCES:

I. R. Juza and K. Bär. Z. anorg. allg. Chem. 283, 230 (1956).

- II. M. von Stackelberg and R. Paulus. Z. phys. Chem. (B) 28, 427 (1935).
- III. H. Krebs, K. H. Müller and G. Zürm. Z. anorg. allg. Chem. 285, 15 (1956).

## Cadmium Arsenides

## Cd<sub>3</sub>As<sub>2</sub>, CdAs<sub>2</sub>

 $3 \text{ Cd} + 2 \text{ As} = \text{Cd}_3 \text{As}_2$   $\text{Cd} + 2 \text{ As} = \text{CdAs}_2$  $337.2 \quad 149.8 \quad 487.1 \quad 112.4 \quad 149.8 \quad 262.2$ 

The preparation of  $Cd_3As_2$  is similar to that of  $Zn_3As_2$ (method I): heating the metal in a stream of hydrogen that carries arsenic vapor (see p. 1083).

The phase diagram indicates the existence of CdAs<sub>2</sub>, which may be prepared by fusing the components.

PROPERTIES:

Cd<sub>3</sub>As<sub>2</sub>: Gray. M.p. 721°C. Hardness <3.5. d (x-ray) 6.35, d (pycn.) 6.211. Crystal structure: tetragonal, type D5<sub>9</sub> (Zn<sub>3</sub>P<sub>2</sub>). CdAs<sub>2</sub>: Gray-black. M.p. 621°C. Hardness 3.5-4.

**REFERENCES**:

M. von Stackelberg and R. Paulus. Z. phys. Chem (B) <u>28</u>, 427 (1935); A. Granger. Comptes Rendus Hebd. Séances Acad. Sci. <u>138</u>, 574 (1904); Zemczuny. Z. Metallographie <u>4</u>, 228 (1913).

#### Diethylcadmium

#### $Cd(C_2H_5)_2$

 $C_2H_5Br + Mg = C_2H_5MgBr$ 109.0 24.3 133.3

Anhydrous, finely ground  $CdBr_2$  (136 g., 0.5 mole) is added in small portions (vigorous shaking and no cooling) to a solution of  $C_2H_5MgBr$  in 350 ml. of absolute ether. The latter reagent is prepared from 29 g. (1.2 moles) of Mg and 131 g. (1.2 moles) of  $C_2H_5Br$ , the amount required to dissolve the metal. The major portion of the ether is distilled off in a stream of N<sub>2</sub> on a water bath whose temperature does not exceed 80 °C. The solid, porous, gray mass left in the flask is then distilled at 1 mm. into a liquid-nitrogen-cooled trap, while the temperature of the oil bath is raised from 20 to 120°C over the one-hour distillation period. The clear, completely colorless distillate is carefully freed of ether by distilling the latter in a nitrogen stream; the residue is distilled in  $N_2$  at 19.5 mm. All the Cd(C<sub>2</sub>H<sub>5</sub>)<sub>2</sub> goes over at 64.0°C; it is analytically pure. Yield 90%.

PROPERTIES:

Colorless oil with an unpleasant odor. M.p.  $-21^{\circ}$ C, b.p. (19.5 mm.) 64°C, (760 mm.) 164.7°C. Decomposes at 150°C, explosively at 180°C. May be stored without decomposition in a sealed tube filled with N<sub>2</sub>. Fumes explosively in air, at first forming white and then (rapidly) brown clouds accompanied by violent detonation. Decomposed by H<sub>2</sub>O with a characteristic crackling sound continuing for hours on end. d (21.7°C) 1.653.

**REFERENCE:** 

E. Krause. Ber. dtsch. chem. Ges. 50, 1813 (1918).

## Cadmium Carbonate

## CdCO<sub>3</sub>

I.	$CdCl_2 +$	$(NH_4)_2CO_3$	$+ 4 \text{ NH}_3 =$	[Cd(NH <sub>3</sub> ) <sub>4</sub> ]	$CO_3 + 2 NH_4Cl$
	183.3	96.1	68.1	240.5	107.0
		[Cd(NH	$[a_3)_4]CO_3 = 0$	$CdCO_3 + 4N$	$\mathbf{M}_{3}$
		240	0.5	172.4 68	3.1

A solution of  $(NH_4)_2CO_3$  is added all at once to a solution of  $CdCl_2$ , followed by the quantity of ammonia necessary to dissolve the resultant precipitate. The liquid is then heated in an open vessel on a water bath. The  $CdCO_3$  separates as shiny crystals.

II. 
$$CdCl_2 + 2 HCl + 3 H_2O + 2 CO(NH_2)_2 = CdCO_3 + 4 NH_4Cl + CO_2$$
  
183.3 72.9 54.1 120.1 172.4 214.0 44.0

A vertical bomb (wall thickness 3 mm., diameter 25 mm., height about 50 cm.) contains a solution of 10 mmoles of  $CdCl_2$  in 30 ml. of H<sub>2</sub>O, 0.3 ml. of conc. hydrochloric acid, and a small glass beaker filled with 20 mmoles of urea. A long stem from the bottom of the bomb supports the beaker above the liquid surface. The bomb is melt-sealed and heated at 200°C for 18-24 hours. The yield is almost quantitative.

PROPERTIES:

White powder or rhombohedral leaflets. Sparingly soluble in  $H_2O$ , soluble in acid. Vapor pressure at decomposition (321°C) 77 mm., (357°C) 760 mm. d 4.258. Crystal structure: type GO<sub>1</sub> (calcite).

**REFERENCES**:

I. A. de Schulten. Bull. Soc. Chim. France [3] <u>19</u>, 34 (1898). II. W. Biltz. Z. anorg. allg. Chem. <u>220</u>, 312 (1934).

## Cadmium Acetate

#### Cd(CH<sub>3</sub>COO)<sub>2</sub>

 $\begin{array}{rll} Cd(NO_3)_2 \cdot 4 H_2O &+ (CH_3CO)_2O &= Cd(CH_3COO)_2 &+ 2 HNO_3 &+ 3 H_2O \\ 308.5 & 102.1 & 230.5 & 124.0 \end{array}$ 

A mixture of 5 g. of  $Cd(NO_3)_2 \cdot 4H_2O$  and 25 ml. of acetic anhydride is heated; when the vigorous reaction has ceased, the mixture is refluxed 15 minutes. After cooling and suctionfiltering, the white, crystalline precipitate is washed with some acetic anhydride and ether, and vacuum-dried over KOH and  $H_2SO_4$ . Yield 3.6 g. (97%).

PROPERTIES:

Colorless crystals. M.p. 254-256°C; d 2.341.

**REFERENCE**:

E. Späth. Monatsh. Chem. <u>33</u>, 241 (1912).

## Cadmium Cyanide

#### Cd(CN)<sub>2</sub>

 $Cd(OH)_2 + 2 HCN = Cd(CN)_2 + 2 H_2O$ 146.4 54.1 164.5 36.0

Evaporation of a solution of  $Cd(OH)_2$  in aqueous HCN precipitates  $Cd(CN)_2$  in the form of crystals. These are dried at 110°C.

PROPERTIES:

Air-stable crystals; turn brown on heating in air. Solubility (15°C) 1.7 g./100 ml.  $H_2O$ . Soluble in KCN solution. d 2.226. Crystal structure: type C 3 (Cu<sub>2</sub>O type).

**REFERENCE:** 

W. Biltz. Z. anorg. allg. Chem. <u>170</u>, 161 (1928).
## Potassium Tetracyanocadmate

## K<sub>2</sub>Cd(CN)<sub>4</sub>

 $\begin{array}{rcl} Cd(CN)_2 \ + \ 2 \ KCN \ = \ K_2 Cd(CN)_4 \\ 164.4 & 130.2 & 294.7 \end{array}$ 

Cadmium cyanide, obtained by precipitation of a  $CdSO_4$  solution with KCN and filtering, is dissolved (shaking) in an aqueous solution of the stoichiometric quantity of KCN. The filtrate is crystallized by evaporation. The product is dried at 105 °C.

SYNONYMS:

Potassium cadmium cyanide, cadmium potassium cyanide.

**PROPERTIES:** 

Octahedral, very refractive, air-stable crystals. Solubility (cold) 33.3 g., (b.p.) 100 g./100 ml.  $H_2O$ ; (20°C) 2 g./100 g. of 88% v./v. alcohol. M.p. about 450°C; d 1.846. Crystal structure: type  $HI_1$  (spinel type).

**REFERENCE:** 

W. Biltz. Z. anorg. allg. Chem. <u>170</u>, 161 (1928).

## Cadmium Thiocyanate

#### $Cd(SCN)_2$

 $\begin{array}{rll} CdSO_4 &+& Ba(SCN)_2 &=& Cd(SCN)_2 &+& BaSO_4 \\ (^{6}\!\!/_8 \,H_2O) \\ 256.5 && 253.5 && 228.6 && 233.4 \end{array}$ 

To a boiling solution of 12.68 g. of Ba(SCN)<sub>2</sub> is added, in drops, 12.83 g. of  $CdSO_4 \cdot {}^8/_3 H_2O$  in 100 ml. of boiling  $H_2O$ , taking care to keep the liquid boiling. After cooling, the precipitate is allowed to settle and the mixture is filtered after standing for some time. The filtrate is evaporated to 80 ml., filtered again, and evaporated to dryness on a water bath.

The required starting solution of Ba(SCN)<sub>2</sub> is prepared by dissolving 15.78 g. of Ba(OH)<sub>2</sub>  $\cdot$  8 H<sub>2</sub>O in 500 ml. of H<sub>2</sub>O and allowing it to react with a solution of 7.62 g. of NH<sub>4</sub>SCN in 100 ml. of H<sub>2</sub>O, according to the equation

 $Ba(OH)_2 \cdot 8 H_2O + 2 NH_4SCN = Ba(SCN)_2 + 2 NH_3 + 10 H_2O$ 315.5 152.2 253.5 34.1 180.2 The mixture is brought to a boil and heated until  $NH_3$  ceases to evolve.

SYNONYMS:

Cadmium rhodanide.

PROPERTIES:

Colorless crystal crusts. Soluble in  $\rm H_2O,$  alcohol and liquid  $\rm NH_3.$ 

**REFERENCE**:

H. Grossmann. Ber. dtsch. chem. Ges. 35, 2666 (1902).

# **Cadmium Silicate**

#### Cd<sub>2</sub>SiO<sub>4</sub>

 $\begin{array}{rrrr} 2\,CdO &+ SiO_2 &= Cd_2SiO_4 \\ 256.8 & 60.1 & 316.9 \end{array}$ 

Like  $Zn_2SiO_4$ ,  $Cd_2SiO_4$  is prepared from the oxides by fusion or by hydrothermal synthesis (see p. 1089).

SYNONYM:

Cadmium orthosilicate.

PROPERTIES:

M.p. 1246°C; d 5.833. Phosphorescent after activation with manganese.

**REFERENCE**:

W. Biltz and A. Lemke. Z. anorg. allg. Chem. <u>203</u>, 330 (1932);
C. J. van Nieuwenburg and H. B. Blumendahl. Rec. Trav. Chim. Pays-Bas <u>50</u>, 989 (1931).

Cadmium Ferrate (III)

CdFe<sub>2</sub>O<sub>4</sub>

 $CdCO_3 + Fe_2O_3 = CdFe_2O_4 + CO_2$ 172.4 159.7 288.1 44.0

Finely screened CdCO<sub>3</sub> [or Cd(OH)<sub>2</sub>] and  $\gamma$ -FeOOH (p. 1500) in a Cd:Fe ratio of 1:2 are mixed for five hours on a mechanical shaker in an atmosphere free of  $CO_2$  and  $H_2O$ . The loose powder is divided into 5-g, portions, which are heated for one hour in a platinum crucible at 800°C [mixtures with Cd(OH)<sub>2</sub> at 1000°C] and then allowed to cool to room temperature in a desiccator. The compound cannot be prepared by precipitation.

PROPERTIES:

Dark brown, hygroscopic powder; stable at room temperature. Crystal structure: type H1<sub>1</sub> (spinel type).

REFERENCE:

R. Fricke and F. Blaschke. Z. anorg. allg. Chem. 251, 396 (1943).

# Mercury (II) Oxychloride

## HgCl<sub>2</sub> · 4 HgO

I. To prepare brown  $HgCl_2 \cdot 4 HgO$ , a solution of 15.0 g. of borax in 1.5 liters of water is added to a solution of 10.0 g. of mercuric chloride in 2.0 liters of water (both solutions are at 50-55°C). The desired compound separates on cooling in the form of thin, flexible flakes 0.1-0.8 mm. long. Depending on the thickness of these flakes, the color of the precipitate varies from golden yellow to black-brown. The yield is about 85%, based on the HgCl<sub>2</sub> used.

II. Black  $HgCl_2 \cdot 4HgO$  is obtained when 5.0 g. of finely crystalline brown  $HgCl_2 \cdot 4HgO$  is heated for 72 hours in a sealed tube with 10 ml. of 0.1N HNO<sub>3</sub> at 180°C. The product separates in the form of rhombic needles several millimeters long, which are readily isolated from the basic nitrate present in the mixture.

SYNONYM:

Mercuric oxychloride.

PROPERTIES:

I. Depending on the thickness, golden yellow to dark brown flakes; elongated brown crystals from concentrated solutions. Soluble in hot water (partial decomposition). The powder pattern distinguishes it from the black form (appearance of lattice defects). Brown  $HgCl_2 \cdot 4HgO$  has a remarkably wide phase range, from 3.82 to 4.00 HgO.

1108

II. Black, needle-shaped crystals, sparingly soluble in water. d (pycn.) 9.01. Crystal structure: orthorhombic, space group  $D_{2h}^{26}$ .

**REFERENCES**:

- I. A. Weiss, G. Nagorsen and A. Weiss. Z. Naturforsch. <u>9b</u>, 81 (1954).
- II. A. Weiss. Private communication.

## Mercury (II) Bromide

#### HgBr<sub>2</sub>

 $Hg + Br_2 = HgBr_2$ 200.6 159.8 360.4

Five parts of Hg are covered with 60 parts of  $H_2O$  and allowed to react at 50°C (vigorous stirring) with four parts of  $Br_2$ , which is added dropwise as long as no permanent color is formed. The solution is then brought to a boil, filtered hot, and placed in an ice bath to induce crystallization. The salt is dried at as low a temperature as possible. Purification is by careful double or triple sublimation from a porcelain dish heated on a sand bath and covered with a Petri dish. When very high purity is required (e.g., for conductivity measurements), it may be necessary to repeat the sublimation several times more.

PROPERTIES:

Colorless, lustrous crystal flakes (sublimate and from  $H_2O$ ), rhombic prisms or needles (from alcohol). Light yellow liquid between m.p. 238 °C and b.p. 320.3 °C. d<sup>20</sup> 5.73; d<sup>242</sup> (liq.) 5.11. Vapor pressure (200 °C) 24.1 mm., (280 °C) 334.2 mm.; sublimes without decomposition. Solubility (25 °C) 0.62 g., (100 °C) 22 g. per 100 g.  $H_2O$ ; (25 °C) 30.0 g./100 g. ethanol; (25 °C) 69.4 g./100 g. methanol. Specific electrical conductivity (242 °C) 1.45  $\cdot$  10<sup>-4</sup> ohm<sup>-1</sup>. Its melt is a solvent for a large number of inorganic and organic substances, which impart considerable conductivity to the melt. Cryoscopic constant 36.7 deg. per mole in 1000 g. of HgBr<sub>2</sub>. Rhombic layered lattice, space group  $C_{2y}^{12}$ .

REFERENCES:

 W. Reinders. Z. phys. Chem. (A) <u>32</u>, 514 (1900); G. Jander and K. Brodersen. Z. anorg. Chem. <u>261</u>, 264 (1950).

#### Potassium Triiodomercurate (II)

#### KHgI<sub>3</sub>·H<sub>2</sub>O

 $KI + HgI_2 + H_2O = KHgI_3 \cdot H_2O$ 166.0 454.5 18.0 638.5

First, 41 g. of KI and 59 g. of HgI<sub>2</sub> are dissolved in 14 (!) ml. of hot H<sub>2</sub>O. The beaker should be tared precisely so that evaporating H<sub>2</sub>O may be replenished. The salt crystallizes in yellow needles from the cooling solution. The above instructions result from a study of the three-component system HgI2-KI-H2O showing that in this system only  $KHgI_3 \cdot H_2O$  can precipitate. Yield 9 g. The salt  $K_2HgI_4$  (without water of crystallization) is obtained

from acetone containing exactly 2% H<sub>2</sub>O (between 34 and 56°C).

Use: A solution treated with KI is known as Thoulet's solution (see Part I, p. 99); it is also a reagent for alkaloids. Alkaline K<sub>2</sub>HgI<sub>4</sub> solution is Nessler's reagent.

SYNONYMS:

Mercury potassium iodide, potassium mercuriiodide, potassium iodohydrargyrate.

PROPERTIES:

Light yellow crystal needles; becomes orange-red in a vacuum desiccator over H<sub>2</sub>SO<sub>4</sub> (reversible loss of H<sub>2</sub>O). Decomposes in H<sub>2</sub>O with loss of HgI<sub>2</sub>. Soluble in KI solution. Sublimes off HgI<sub>2</sub> on heating.

REFERENCES:

M. Pernot. Comptes Rendus Hebd. Séances Acad. Sci. 182, 1154 (1926); <u>185</u>, <u>950</u> (1927); Ann. Chim. [10] <u>15</u>, 5 (1931).

## Copper (I) Tetraiodomercurate (II)

#### Cu<sub>2</sub>HgI<sub>4</sub>

I.

$2  CuSO_4$	+	K₂HgI₄	+	$SO_2$	+	$2 H_2O$
$(5 H_2 O)$						
499.4		786.5		64.1		36.0

 $= Cu_2HgI_4 + K_2SO_4 + 2H_2SO_4$ 835.4 174.3196.2

A solution of 4.5 g. of HgI<sub>2</sub> and 3.3 g. of KI in 25 ml. of H<sub>2</sub>O is filtered and treated with a solution of 5 g. of  $CuSO_4 \cdot 5 H_2O$  in 15 ml. of H<sub>2</sub>O; SO<sub>2</sub> is then introduced. The resultant bright-red precipitate is suction-filtered, washed with  $H_2O$ , and dried at 100°C. The compound may be recrystallized from hot hydrochloric acid.

II.

$$2 \text{ CuI} + \text{HgI}_2 = \text{Cu}_2 \text{HgI}_4$$
  
380.9 454.5 835.4

The components are mixed in stoichiometric proportions and fused over an open flame in an evacuated Pyrex glass bomb, which should be as small as possible. Pure  $Cu_2HgI_4$  is obtained.

An analogous procedure yields Ag<sub>2</sub>HgI<sub>4</sub>.

PROPERTIES:

Red crystalline powder or small, tabular crystals. d 6.094. Crystal structure: tetragonal. Space group  $D_{2d}^{11}$ ; on heating to 70°C, the color changes to chocolate brown with simultaneous enantiotropic structure transformation to cubic type B3 (zincblende) in which the cation distribution is random.

REFERENCES:

- I. Cabentou and Willm. Bull. Soc. Chem. France [2] <u>13</u>, 194 (1870);
   J. A. A. Ketelaar. Z. Kristallogr. <u>80</u>, 190 (1931).
- II. H. Hahn, G. Frank and W. Klingler. Z. anorg. allg. Chem. <u>279</u>, 271 (1955).

# Mercury (II) Sulfide

HgS

BLACK MODIFICATION

$$HgCl_2 + H_2S = HgS + 2 HCl$$
  
271.5 22.4 l. 232.7 72.9

Hydrogen sulfide is introduced into a mercury (II) solution in 1-2N HCl, absolutely free of oxidizing agents. The transient white to brownish precipitate reacts with additional  $H_2S$  to yield black HgS, e.g.,

$$Hg_3S_2Cl_2 + H_2S = 3 HgS + 2 HCl.$$

SYNONYMS:

Ethiop's mineral; mercuric sulfide, black.

RED MODIFICATION

$$Hg(CH_{3}COO)_{2} + H_{2}S = HgS + 2 CH_{3}COOH$$
  
318.7 22.4 l. 232.7 120.1

A solution of 35 g. of  $Hg(CH_3COO)_2$  and 25 g. of  $NH_4SCN$  in 100 ml. of hot glacial acetic acid is prepared. A moderately fast stream of  $H_2S$  is then introduced until precipitation is complete. The acetic acid is then slowly evaporated (caution, HCN!), and the black precipitate transforms to the red form. The glacial acetic acid must be present until conversion is complete; overheating must be avoided. During the last stage the paste must be constantly stirred. If this is neglected, the product is dull red or brown. When the acid has been completely removed and the product cooled, 200 ml. of  $H_2O$  is added and the mixture is filtered through a Büchner funnel. The product is washed and dried between two layers of thick filter paper. Yield 25 g.

If HgCl<sub>2</sub> is used instead of the acetate, a larger amount of glacial acetic acid is necessary and the boiling must be longer. The final color, however, is never as magnificent as that of the product prepared from the acetate.

Alternate methods: a) A conc. solution of  $HgCl_2$  (20 ml.) is poured into 12 ml. of aqueous ammonia (1:2). The resultant precipitate is treated with a somewhat larger quantity of conc.  $Na_2S_2O_3$ solution than is necessary for complete solution. The mixture is heated in a dish and evaporated until pasty; the paste is filtered and washed with hot  $H_2O$ .

b) A dry method for the preparation of cinnabar consists in subliming black HgS, grinding the sublimate under  $H_2O$ , freeing it from excess sulfur by boiling with a solution of  $K_2CO_3$ , washing and drying at 70°C.

SYNONYMS:

Cinnabar, vermilion, Chinese red, cinnabarite.

PROPERTIES:

Black modification: Velvety black amorphous powder (tetrahedral crystals). Soluble in aqua regia and in conc. solutions of alkali sulfides, forming thio salts. Unstable. d (x-ray) 7.69. The mineral metacinnabarite crystallizes as type B3 (zincblende).

Red modification: Scarlet powder, darkens in air. Soluble in aqua regia, less readily soluble than the unstable modification in alkali sulfide solutions. Sublimes at 580°C; d 8.09. Hardness 2-2.5. Hexagonal, deformed NaCl lattice, type B9 (cinnabarite type).

1112

**REFERENCES:** 

L. C. Newell, R. N. Maxson and M. H. Filson in: H. S. Booth. Inorg. Syntheses, Vol. I, New York-London, 1939, p. 19; O. Hausamann. Ber. dtsch. chem. Ges. 7, 1747 (1874).

# Mercury (II) Selenide HgSe

I.

$$HgCl_2 + H_2Se = HgSe + 2 HCl$$
  
271.5 81.0 279.6 72.9

A dilute solution of  $HgCl_2$  is added in drops (stirring) to a saturated aqueous solution of  $H_2Se$ , so that  $HgCl_2$  is never present in excess. Air is carefully excluded. If a conc. solution of  $HgCl_2$  is used, the precipitate formed consists of yellow  $HgCl_2 \cdot 2 HgSe$ .

II.  $\begin{array}{rcrr} \text{HgCl}_2 &+ 2 \text{ NaCN} &+ 8 \text{ NH}_4\text{OH} &+ \text{SeO}_2 &+ 3 \text{ SO}_2 \\ &&&&&\\ 271.5 & 98.0 & 280.4 & 101.0 & 192.2 \\ \\ &= \text{HgSe} &+ 2 \text{ NaCl} &+ 2 \text{ NH}_4\text{CN} &+ 3 (\text{NH}_4)_2\text{SO}_4 &+ 4 \text{ H}_2\text{O} \\ &&&&\\ 279.6 & 116.9 & 88.1 & 396.4 & 72.1 \end{array}$ 

Mercury (II) chloride is added to an equivalent amount of NaCN solution; the mixture is made strongly alkaline with conc. ammonia and an equivalent amount of  $SeO_2$  is added. The mixture is filtered and  $SO_2$  is introduced. The liquid must be maintained alkaline to prevent precipitation of red selenium. The end of the neutralization may be spotted by the reduced rate of absorption of SO<sub>2</sub>. The black HgSe is suction-filtered, washed with a dilute ammoniacal NaCN solution followed by H<sub>2</sub>O, and dried in a desiccator over  $P_2O_5$ .

III.	Hg +	Se =	HgSe
	200.6	79.0	279.6

Stoichiometric quantities of Hg and Se are heated to 550-600 °C in a sealed bomb.

The product may be purified by sublimation at 600-650  $^{\circ}\mathrm{C}$  in a stream of very pure N<sub>2</sub>.

SYNONYM:

Mercuric selenide.

PROPERTIES:

Gray-black (the sublimed material is violet-black) regular crystals with a metallic luster. Sublimes without decomposition at about 600 °C in N<sub>2</sub>, CO<sub>2</sub> or vacuum. Soluble in NH<sub>4</sub> HSe, giving a red solution. d 8.266. Hardness (of tiemannite) 2.5. Crystal structure: type B3 (zincblende type).

**REFERENCES:** 

- I. L. Moser and K. Atynski. Monatsh. Chem. 45, 235 (1925).
- II. H. Hahn and G. Störger. Unpublished data.
- III. G. Pellini and R. Sacerdoti. Gazz. Chim. Ital. <u>40</u>, II, 42 (1910); Chem. Zentralbl. <u>1910</u>, II, 1741.

# Mercury (II) Amide Chloride

# HgNH<sub>2</sub>CI

A solution of 20 g. of  $HgCl_2$  in 400 ml. of  $H_2O$  is mixed with 31 ml. of 6N (10%) ammonia. The resultant precipitate is allowed to settle and is then suction-filtered and washed with 180 ml. of cold water. This amount of wash water must be adhered to precisely, since it affects the composition of the product: with larger quantities of  $H_2O$  the product assumes a yellow color due to partial formation of  $NHg_2Cl \cdot H_2O$ . The product is dried at 30°C (exclusion of light); when it appears to be dry, it is ground and dried further at 30°C. Yield 18.5 g.

SYNONYMS:

White mercuric precipitate (infusible); ammoniated mercuric chloride.

PROPERTIES:

White, light-sensitive powder. Insoluble in  $H_2O$ ; decomposes in  $H_2O$  and alcohol. Completely soluble in  $CH_3COOH$ . Does not melt on heating, but volatilizes with decomposition. d 5.38.

REFERENCES:

E. Mannerheim. Pharmazeutische Chemie [Pharmaceutical Chemistry], IV, Exercise Compounds, Collection, <u>682</u>, p. 63 (1921);
 J. Sen. Z. anorg. allg. Chem. <u>33</u>, 197 (1903).

# Diamminemercury (II) Dichloride

# HgCl<sub>2</sub> · 2 NH<sub>3</sub>

$$HgCl_{2} + 2 NH_{3} = HgCl_{2} \cdot 2 NH_{3}$$
  
271.5 34.1 305.6

A solution of 5 g. of  $HgCl_2$  and 3 g. of  $NH_4Cl$  in 100 ml. of  $H_2O$  is allowed to react with 20 ml. of 4.5N (8%) ammonia. The

1114

mixture is left to stand six days (frequent shaking). The resultant precipitate consists of small, colorless crystals. It is washed with alcohol and dried in the dark over KOH.

II. In the absence of moisture, the product of theoretical composition is formed in 1.5 days via addition of  $NH_3$  to  $HgCl_2$  at room temperature in a vapor pressure eudiometer (cf. Part I, p. 102). The other known ammines are  $HgCl_2 \cdot 8NH_3$  and  $HgCl_2 \cdot 9.5 NH_3$ .

SYNONYM:

White mercuric precipitate (fusible).

PROPERTIES:

Fine rhombic dodecahedra comprise the crystalline, air-stable, white powder. Melting range 247 to 253 °C under NH<sub>3</sub> at atmospheric pressure. Soluble in CH<sub>3</sub>COOH. Stable as a precipitate in a solution containing more than 1.7 g. of NH<sub>2</sub>Cl/100 ml. of H<sub>2</sub>O. Melts with decomposition on heating. d 3.77. Crystal structure: cubic with random distribution of Hg.

**REFERENCES:** 

I. D. Strömholm. Z. anorg. allg. Chem. <u>57</u>, 86 (1908).

II. W. Biltz and C. Mau. Ibid. <u>148</u>, 170 (1925).

Mercury (II) Iminobromide

#### Hg<sub>2</sub>(NH)Br<sub>2</sub>

I. A solution of 2.16 g. of  $HgBr_2$  in 80 ml. of boiling water is mixed with a solution of 0.2 g. of  $NH_4Br$  in 100 ml. of 0.1N ammonia. A yellow precipitate forms immediately; it is suction-filtered while still hot, washed with 250 ml. of cold water, and dried over NaOH. Yield 1.4 g. (81%).

II. A concentrated solution of 1.44 g. of  $HgBr_2$  and 0.9 g. of  $NH_4Br$  (total volume about 30 ml.) is added to 2.6 g. of freshly precipitated HgO. The mixture is shaken for about six hours; it clarifies after 2.5 hours. The precipitate is worked up as in method I.

SYNONYM:

Mercuric iminobromide.

PROPERTIES:

Yellow, light-sensitive powder. Soluble in KCN and KI solutions, insoluble in organic solvents. Shaking with cold, aqueous ammonia yields the bromide of Millon's base (see below).

REFERENCES:

- I. W. Rüdorff and K. Brodersen. Z. anorg. allg. Chem. 270, 145 (1952).
- II. A. Meuwsen and G. Weiss. Ibid. 289, 5 (1957).

## Millon's Base

#### $NHg_2OH \cdot x H_2O (x = 1 \text{ or } 2)$

2 HgO	+	NH₄OH	=	$NHg_2OH \cdot 2 H_2O$	<u>110 °C</u>	$NHg_{2}OH \cdot H_{2}O$
433.2		35.1		468.3		450.2

Freshly precipitated HgO (see below) is taken up in carbonatefree, approximately 12N ammonia, and shaken in the dark for 14 days. The resultant light-yellow microcrystalline precipitate is filtered and washed with some water. It is dried over silica gel in a desiccator.

This dihydrate of Millon's base may be converted to the brown monohydrate by brief drying (10 minutes at 110°C). This compound is stable in vacuum over silica gel.

The starting HgO is precipitated at 70 °C by addition of a solution of 7.5 g. of NaOH in 20 ml. of  $H_2O$  to a solution containing 25 g. of HgCl<sub>2</sub> in 200 ml. of  $H_2O$ .

The starting ammonia solution is prepared by passing 51 g. of  $CO_2$ -free NH<sub>3</sub> through 250 ml. of boiled distilled water.

PROPERTIES:

a) Dihydrate: Very fine, yellow, hexagonal crystals, lightsensitive, insoluble in alcohol and ether. Converted to the monohydrate above 110°C. d (x-ray) 7.33. Crystal structure: hexagonal.

b) Monohydrate: Brown, light-sensitive powder. In moist air, the color changes to yellow, with formation of the dihydrate. d (x-ray) 7.05. Crystal structure: hexagonal.

**REFERENCE:** 

W. Rüdorff and K. Brodersen. Z. anorg. allg. Chem. 274, 338 (1953).

1116

## Bromide of Millon's Base

#### NHg<sub>2</sub>Br

Ammonia (5 ml., 24%) is diluted with 400 ml. of water. A solution of  $HgBr_2$  in 200 ml. of water, saturated at 20°C, is added with stirring. The resultant yellow precipitate is filtered and washed with water until the filtrate is free of bromide ion. The light yellow product is dried at 110°C and stored in vacuum over silica gel.

The iodide and the nitrate of Millon's base are also readily prepared (see reference below).

PROPERTIES:

Light yellow, finely grained powder. d (pycn.) 7.64, d (x-ray) 7.66. Crystal structure: hexagonal (space group  $D_{5h}^4$ ).

REFERENCE:

W. Rüdorff and K. Brodersen. Z. anorg. allg. Chem. <u>274</u>, 338 (1953).

#### Mercury (I) Thionitrosylate

#### $[Hg_2(NS)_2]_x$

$2 \operatorname{Hg}_2(\operatorname{NO}_3)_2$	+	$S_4(NH)_4$	=	$2 \operatorname{Hg}_2(\operatorname{NS})_2$	+	4 HNO <sub>3</sub>
$(2 H_2O)$						
1122.6		188.3		986.7		252.1

A solution of 2.3 g. of  $Hg_2(NO_3)_2$  (carefully predried over  $P_2O_5$ ) in about 60 ml. of dimethylformamide is prepared. Any basic nitrate which may precipitate is filtered off. The solution is cooled in a  $-70^{\circ}$ C bath and 400 mg. of  $S_4(NH)_4$ , dissolved in a few milliliters of dimethylformamide, is added. The resultant yellow solution is slowly brought to room temperature. A yellow precipitate begins to separate at about 0°C. The mixture is allowed to stand 10 minutes and is filtered on a coarse fritted glass filter. The precipitate is washed with dimethylformamide, followed by acetone. The product is dried in high vacuum, first at room temperature, then for a short time at 100°C. Yield 1-1.5 g.

SYNONYM:

Mercurous thionitrosylate.

PROPERTIES:

Solid yellow substance. Insoluble in all common solvents. Stable at room temperature; detonates when held in a flame. Hydrolyzed by bases, evolving  $NH_3$ ; reacts with strong acids, forming basic Hg salts,  $SO_2$  and ammonium salts.

**REFERENCE:** 

M. Goehring and G. Zirker. Z. anorg. allg. Chem. 285, 70 (1956).

# Mercury (II) Thionitrosylate

## [Hg(NS)<sub>2</sub>]<sub>x</sub>

A solution of 27 g. of Hg (CH<sub>3</sub>COO)<sub>2</sub> in absolutely dry pyridine is prepared. Then 1 g. of S<sub>4</sub>(NH)<sub>4</sub> in 20 ml. of pyridine is added, resulting in the appearance of a blood-red color, followed soon by separation of a fine-grained yellow precipitate. The mixture is mechanically shaken until the supernatant becomes pure yellow: this requires about two hours. During this operation, the temperature should be maintained at 20-25°C; the reaction is inhibited at lower temperatures. After the shaking, the deposit is washed several times with pyridine (total 80 ml.) and the solvent is decanted. The remaining pyridine is removed by ether extraction. The product is vacuum-dried over conc. H<sub>2</sub>SO<sub>4</sub>. Yield 2 g. (65%).

PROPERTIES:

Finely divided yellow powder. Insoluble in all common solvents. In water, decomposes with blackening. Stable at room temperature, decomposes at about 140°C.

SYNONYM:

Mercuric thionitrosylate.

REFERENCE:

A. Meuwsen and M. Lösch. Z. anorg. allg. Chem. 271, 217 (1953).

## Diethylmercury

 $Hg(C_2H_5)_2$ 

Magnesium turnings (25 g.) are covered with 500 ml. of dry ether in a two-liter three-neck flask provided with a dropping funnel and reflux condenser. Ethyl bromide (125 g.) is gradually added in drops from a funnel. The rate of the reaction, which starts after a few minutes, is regulated by the rate of addition. It may be slowed by cooling with water. When the reaction ceases, the solution is boiled for about 30 minutes, cooled and filtered through glass wool.

The ethereal  $C_2H_5MgBr$  solution is treated portionwise (stirring) with 97 g. of HgCl<sub>2</sub> in a second three-neck two-liter flask, equipped with a stirrer and a reflux condenser. The addition is spread out over 45 minutes to avoid a too violent reaction. The solution is then boiled for about 10 hours, after which 250 ml. of water is slowly added through the condenser to hydrolyze the excess  $C_2H_5$ -MgBr. The ether layer is separated and dried over CaCl<sub>2</sub>. After removal of the ether by distillation, the residue is distilled under reduced pressure. The Hg( $C_2H_5$ )<sub>2</sub> goes over between 97 and 99°C at 100 mm. Yield 55 g. (~60%).

Dimethylmercury may be obtained in a similar manner according to

 $2 \operatorname{CH}_3 \operatorname{MgBr} + \operatorname{HgCl}_2 = \operatorname{Hg}(\operatorname{CH}_3)_2 + \operatorname{MgBr}_2 + \operatorname{MgCl}_2.$ 

Storage ampoules are first filled with  $N_2$  or  $CO_2$  to avoid explosions during melt-sealing. To avoid the unpleasant results of shattering of the ampoules on explosion (flying glass, etc.), the ampoules should be stored in cotton wool inside well-stoppered powder jars.

PROPERTIES:

Almost odorless, heavy liquid. B.p. (760 mm.) 159°C. Stable to  $H_2O$  and air. Gradually decomposes when stored in light (Hg drops appear). Almost insoluble in  $H_2O$ , sparingly soluble in alcohol, soluble in ether. d (20°C) 2.466. Very toxic.

The danger of poisoning is especially great when the compound is spilled on a porous surface, such as a laboratory bench or wooden floor. The material is inactivated by hot hydrochloric acid or hot mercuric chloride solution (formation of  $C_2H_5HgCl$ ).

REFERENCES:

C. S. Marvel and V. L. Gould. J. Amer. Chem. Soc. 44, 153 (1922); E. Krause and A. von Grosse. Chemie der metallorgan. Verbindungen [Chemistry of Oranometallic Compounds], Berlin, 1937.

#### Mercury (I) Acetate

#### $Hg_2(CH_3COO)_2$

A solution of 20 g. of  $Hg_2(NO_3)_2$  in 120 ml. of water plus 3.5 ml. of 25% HNO<sub>3</sub> is treated with a solution of 15 g. of sodium acetate in 50 ml. of  $H_2O$ . The resultant precipitate is washed with cold water and dried in a desiccator over CaCl<sub>2</sub>. Yield 13 g.

SYNONYM:

Mercurous acetate.

PROPERTIES:

White, light-sensitive crystal flakes (tinged with gray). Solubility  $(15^{\circ}C) 0.75 \text{ g.}/100 \text{ ml. H}_{2}O$ . On boiling and in light the compound in solution disproportionates to Hg and Hg (II) acetate; the mercuric salt then hydrolyzes to a yellow, insoluble basic salt. Readily soluble in dilute acetic acid, insoluble in alcohol and ether. Decomposes on heating, forming a residue of black flakes.

REFERENCE:

Ullmann. Enzykopädie d. techn. Chemie [Encyclopedia of Ind. Chemistry], 2nd ed., IV, 679.

#### Mercury (II) Acetate

## Hg(CH<sub>3</sub>COO)<sub>2</sub>

 $\begin{array}{rcl} HgO &+ & 2 CH_{3}COOH &= & Hg(CH_{3}COO)_{2} &+ & H_{2}O \\ 216.6 & & 120.1 & & 318.7 & & 18.0 \end{array}$ 

A solution of 20 g. of yellow HgO in 30 ml. of 50% CH<sub>3</sub>COOH is prepared on a water bath. It is filtered through a jacketed filter heated with hot water, and the filtrate is cooled with ice. The crystals are suction-dried and washed with ethyl acetate. The product is recrystallized from hot ethyl acetate or from hot water slightly acidified with acetic acid. The salt is dried in a vacuum desiccator over CaCl<sub>2</sub>.

Use: As a mercurizing and oxidizing agent and for the absorption of ethylene. SYNONYM:

Mercuric acetate.

PROPERTIES:

Nacreous, light-sensitive crystalline flakes. On storage acquires a yellow tinge and an odor of  $CH_3COOH$  (formation of a basic salt). M.p. 178-180°C, decomposes at higher temperatures. Solubility (0°C) 25 g., (19°C) 36.4 g./100 ml. H<sub>2</sub>O (and about 100 g. at 100°C with partial dec.). The compound in 0.2N aqueous solution is approximately 30% hydrolyzed; the yellow basic salt precipitates on diluting or heating; Soluble in ethyl acetate. d<sup>23</sup> 3.286.

**REFERENCE:** 

Gmelin-Kraut. Hdb. anorg. Chem. [Handbook of Inorg. Chem.], 7th ed., V2, 826, Heidelberg, 1914, modified.

## Mercury (II) Cyanide

#### Hg(CN)<sub>2</sub>

I.	$HgO \ +$	2  HCN =	$Hg(CN)_2$	+	$H_2O$
	216.6	54.1	252.7		18.0

The crystalline material is obtained by evaporation of a solution of HgO in aqueous HCN. The product is recrystallized, dried at 50°C, ground and redried.

П.	9 HgO +	$Fe_4[Fe(CN)_6]_3$	$+ 9 H_2 O =$
	<sup>1</sup> / <sub>10</sub> 195.0	85.9	16.2
	9 Hg(CN) <sub>2</sub> 227.4	+ 4 Fe(OH) <sub>3</sub> 42.8	$+ 3 \operatorname{Fe}(OH)_2 27.0$

One part of HgO is digested for a few hours on a water bath with one part of Prussian blue and 10 parts of  $H_2O$ . The crystals separate on evaporation of the solution.

SYNONYM:

Mercuric cyanide.

PROPERTIES:

Colorless, prismatic, tetragonal crystals. Decomposes into Hg and  $(CN)_2$  at 320°C. Solubility (0°C) 8 g., (100°C) 53.9 g./100 ml.

H<sub>2</sub>O; (19.5°C) 10.1 g./100 g. ethanol; (19.5°C) 44.1 g./100 g. methanol. d 3.996. Crystal structure: type F1<sub>1</sub> [Hg(CN)<sub>2</sub> type].

**REFERENCES:** 

- I. W. Biltz. Z. anorg. allg. Chem. 170, 161 (1928).
- II. Gmelin-Kraut. Hdb. anorg. Chem. [Handbook of Inorg. Chem.], 7th ed. V2, 832, Heidelberg, 1914.

## Potassium Tetracyanomercurate (II)

#### K<sub>2</sub>Hg(CN)<sub>4</sub>

 $\begin{array}{rcl} Hg(CN)_2 &+& 2 \ KCN &=& K_2 Hg(CN)_4 \\ 252.7 & 130.2 & 382.9 \end{array}$ 

I. A solution of stoichiometric quantities of  $Hg(CN)_2$  and KCN is evaporated to induce crystallization. The product is recrystallized and dried at 80 °C.

II. Treatment of a suspension of  $Hg(CN)_2$  in liquid HCN with the stoichiometric quantity of KCN produces  $KHg(CN)_3$ , in addition to  $K_2Hg(CN)_4$ .

SYNONYM:

Potassium mercuricyanide.

**PROPERTIES:** 

Colorless, octahedral crystals. Solubility (20°C) 1 g./35 g. of 88% v./v. alcohol. d 2.420. Crystal structure: type H1<sub>1</sub> (spinel type).

REFERENCES:

- I. W. Biltz. Z. anorg. allg. Chem. <u>170</u>, 161 (1928).
- II. G. Jander and B. Grüttner. Ber dtsch. chem. Ges. 81, 118 (1948).

## Mercury (I) Thiocyanate

## Hg<sub>2</sub>(SCN)<sub>2</sub>

A slightly acid solution of  $Hg_2(NO_3)_2$ , freed of mercuric ions by means of metallic Hg, is treated with somewhat less than the

1122

stoichiometric amount (~75%) of KSCN solution. A dark-gray salt precipitates at first; after standing for several days (repeated stirring), it becomes completely white. It is washed several times with boiling  $H_2O$ .

SYNONYMS:

Mercurous thiocyanate; mercury (I) rhodanide.

PROPERTIES:

Colorless, light-sensitive powder. Insoluble in  $H_2O$ , soluble in KSCN solution, precipitating Hg. Decomposes on heating, forming a foamy mass. d 5.318.

**REFERENCE:** 

K. Huttner and S. Knappe. Z. anorg. allg. Chem. 190, 27 (1930).

# Mercury (II) Thiocyanate

#### Hg(SCN)<sub>2</sub>

 $\begin{array}{rrrr} Hg(NO_3)_2 &+ 2 \text{ KSCN} &= Hg(SCN)_2 &+ 2 \text{ KNO}_3 \\ (H_2O) \\ 342.6 & 194.3 & 316.8 & 202.2 \end{array}$ 

A Hg(NO<sub>3</sub>)<sub>2</sub> solution, acidified with a few drops of HNO<sub>3</sub>, is treated with the stoichiometric amount of KSCN solution. The resultant crystalline precipitate is suction-filtered and washed with H<sub>2</sub>O. The product may be recrystallized from hot H<sub>2</sub>O or alcohol. Yield 80%.

SYNONYMS:

Mercuric thiocyanate, mercuric sulfocyanate.

PROPERTIES:

Colorless, fibrous needles or nacreous flakes. Solubility  $(25 \,^{\circ}\text{C})$  0.069 g./100 ml. H<sub>2</sub>O. The solubility in alcohol and boiling H<sub>2</sub>O and in KSCN solution is higher, in ether lower. Decomposes with swelling on heating to 165  $^{\circ}\text{C}$ .

REFERENCE:

W. Peters. Z. anorg. allg. Chem. 77, 157 (1912).

# Potassium Tetrathiocyanomercurate (II)

#### K<sub>2</sub>Hg(SCN)<sub>4</sub>

 $\begin{array}{rrrr} Hg(SCN)_2 &+& 2\,KSCN &=& K_2Hg(SCN)_4\\ 316.8 & & 194.3 & 511.1 \end{array}$ 

A boiling solution of 20 g. of KSCN in 100 ml. of  $H_2O$  is mixed with 31.7 g. of Hg(SCN)<sub>2</sub>. The HgS precipitating on cooling is filtered off. The filtrate is concentrated on a water bath until crystallization. The solution then solidifies on further cooling to a white, fibrous crystalline mass. It is suction-filtered and dried over  $P_2O_5$ .

SYNONYM:

Potassium mercurithiocyanate.

PROPERTIES:

Brilliant white crystal needles. Readily soluble in cold water, soluble in alcohol, insoluble in absolute ether.

REFERENCE:

A. Rosenheim and R. Cohn. Z. anorg. allg. Chem. 27, 285 (1901).

# SECTION 21

Scandium, Yttrium, Rare Earths\*

K. WETZEL

#### Pure Scandium Compounds

Scandium may be freed of accompanying elements by extraction of its thiocyanate with ether.

#### A) PURIFICATION OF SMALL QUANTITIES

One gram of the oxide (which should contain as little Ti, Zr and Hf as possible) is dissolved in dilute hydrochloric acid. The solution is evaporated on a water bath until a moist crystal paste is obtained (see note, p. 1126). This is taken up in 60 ml. of 0.5N HCl. Then 53 g. of NH<sub>4</sub>SCN is added (the final volume should be about 100 ml.) and the mixture is shaken with 100 ml. of ether. If a separatory funnel is used, complete phase separation is often diffi-

cult to achieve since the stopcock may become plugged with solid decomposition products of HSCN during the removal of the bottom phase. It is therefore advisable to use a flask such as that in Fig. 284, which has a ground glass stopper at the top, and to use a vacuum in order to transfer the top (or ether) phase into flask c via tube b. Dilute HCl (5-10 ml.) is added to the separated top phase, the ether is evaporated, and the dry residue is treated on a water bath with conc. nitric acid, added in drops (caution! violent reaction). The mixture is then boiled with some additional conc. nitric acid until the orange red HSCN decomposition products disappear. The solution





<sup>\*</sup>In the following text, the rare earths are designated by the general symbol Ln.

is diluted with water and pure scandium is precipitated with dilute ammonia.

This procedure almost completely removes Mg, Ca, Y, the lanthanides, Th and Mn; it also frees the scandium, to a large extent, of Ti, Zr, Hf, U and Fe. However, the product is still contaminated with varying amounts of Be, Al, In, Mo, Re, Fe and Co.

## NOTE:

Very impure raw material samples often yield a noncrystallizing sirup on evaporation of the HCl solution; this sirup should not be heated too long because there is a danger of extensive hydrolysis. A small excess of HCl does not interfere at this stage.

# B) PURIFICATION OF LARGE QUANTITIES

The crude oxide (60 g.) is dissolved in hydrochloric acid and evaporated carefully on a water bath until a moist crystalline mass is formed.

(If the mass should become sirupy instead of crystalline, the evaporation is discontinued and the mixture is diluted in 400-500 ml. of water. Then dilute ammonia is carefully added until the yellow end-point of tropeolin 00, 30 ml. of 2N HCl is added, and the volume is made up to 600 ml. with water. The purification procedure is then continued as described below.)

The paste is dissolved in 600 ml. of 0.1N HCl. If Ti or considerable quantities of Zr and Hf are initially present, hydrolysis products of these elements may still remain; they are filtered off before the next step. The solution is then allowed to react with 500-550 g. of NH<sub>4</sub>SCN and shaken with one liter of ether in a three-liter flask (not a separatory funnel). Just as in method (A), as much of the ether layer as possible is transferred to a second flask containing 100 ml. of a saturated aqueous solution of NH<sub>4</sub> SCN. The acid content of the ether layer, which reaches 0.06N (in HSCN) in the first flask, is largely neutralized by slow addition of 27 ml. of 2N ammonia (the flask is vigorously shaken during this addition). The ether is then shaken in a third and fourth flask each time with 100 ml. of 45%NH<sub>4</sub>SCN solution. The Sc is obtained from this purified ether phase by extraction, in a separatory funnel, with pure water (portions of 250-500 ml.). The extraction is continued until the aqueous layer ceases to yield a precipitate on addition of dilute ammonia (any iron in the starting material will concentration in the first 250-500 ml. of aqueous extract). About 2-3 liters of water is necessary to extract 40-50 g. of Sc<sub>2</sub>O<sub>3</sub> from one liter of ether. The aqueous phase remaining in the first flask is now acidified with 30 ml. of 2N HCl and the operation with one liter of ether is repeated (reuse the aqueous solutions in flasks 2 to 4.) A second repetition of the extraction procedure yields the last traces of Sc.

Such a purification of an oxide initially containing 75-80%  $Sc_2O_3$ , 8-9%  $ZrO_2$ , 0.8-0.9%  $HfO_2$ , 1-2%  $Y_2O_3$ , 0.5%  $Dy_2O_3$ , 1%  $Er_2O_3$ , 0.5%  $Tm_2O_3$ , 6-8%  $Yb_2O_3$ , and 1%  $Lu_2O_3$  gave a 90% yield of  $Sc_2O_3$ . Spectroscopic analysis (x-ray) revealed no other rare earth impurities nor Zr and Hf in this product (limit of detection: 0.1%); the remaining 10% of the scandium oxide present in the raw material was also obtained in greatly concentrated form. After conversion to the oxide, the residue left in the first flask contained less than 0.5%  $Sc_2O_3$ .

Alternate methods: a) Fractional condensation of the chlorides.

b) Fractional sublimation of the acetylacetonates.

c) According to Vickery, pure Sc compounds may be obtained by ion exchange.

#### REFERENCES:

W. Fischer and R. Brock. Z. anorg. allg. Chem. <u>249</u>, 168 (1942); this paper also reexamines several other procedures for extraction of scandium; R. C. Vickery. J. Chem. Soc. (London) <u>1955</u>, 245.

# Treatment of Monazite Sand

Monazite is the orthophosphate of the cerium group of rare earths; it contains oxides of the cerium group (50-70%), oxides of the Y group (1-4%) ThO<sub>2</sub> (1-20%), varying quantities of ThSiO<sub>4</sub>, and small amounts of SiO<sub>2</sub>, Fe<sub>2</sub>O<sub>3</sub> and Al<sub>2</sub>O<sub>3</sub>.

A) EXTRACTION WITH SULFURIC ACID

The monazite sand is ground in a ball mill (final product approx. U.S. 30 mesh). Conc.  $H_2SO_4$  (3.25 kg.) is heated to 200°C in a sixliter porcelain dish, and 3.5 kg. of ground sand is gradually added in small portions (efficient stirring is necessary). Heating and stirring are continued for 30 minutes after completion of the addition, until a dark gray, quite firm paste is obtained.

The paste is slowly poured (stirring) into 25 liters of cold water, and stirring is continued for one hour. If the solution is still warm after this time, ice is added until the temperature drops below 25 °C, since the rare earth sulfates are more soluble in cold than hot water. The residue is allowed to settle, the clear supernatant is decanted, and the solid is suction-filtered and washed several times with cold water. The residue consists of SiO<sub>2</sub>, TiO<sub>2</sub>, ZrO<sub>2</sub> and unreacted monazite. The residues of several extractions may be combined and subjected to a second  $H_2SO_4$  treatment. B) PRECIPITATION OF Th IN THE FORM OF ThP<sub>2</sub>O<sub>7</sub>

The filtrate, which contains  $H_2SO_4$ ,  $H_3PO_4$ , Th and the rare earths, is diluted to 168 liters and stirred for one hour in an earthenware or wooden vessel. The nascent, slightly blue-gray, heavy, gelatinous precipitate is allowed to settle for 8-12 hours. It consists of ThP<sub>2</sub>O<sub>7</sub> contaminated with phosphates of cerium and of other rare earths. The precipitation is complete when no additional solid separates on further dilution. The ThP<sub>2</sub>O<sub>7</sub> is filtered off and washed with water.

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C) PRECIPITATION OF THE RARE EARTHS IN THE FORM OF \ln_2SO_4 \cdot \ln_2(SO_4)_3 \cdot 2H_2O OR \ln_2(C_2O_4)_3 \cdot 5H_2O
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The rare earths are isolated from the filtrate by agitation in the presence of finely ground  $Na_2SO_4$  until the absorption bands of Nd are no longer observable through a 5-cm. layer of the clear supernatant. The precipitate is filtered off, washed and dried at 110°C [if the starting material contains large amounts of yttria earths (i.e., xenotime, YPO<sub>4</sub>) it is best to precipitate with solid oxalic acid, since the double sodium yttria sulfates have appreciable solubilities in water].

A first crude separation of the ceria earths from the yttria earths is achieved by adding solid oxalic acid to the filtrate from the double sulfate precipitation. Use of a saturated solution of oxalic acid yields a precipitate which is difficult to filter and necessitates further dilution of the solution.

# D) CONVERSION OF THE SULFATES OR OXALATES TO THE OXIDES

The double sulfates or oxalates are mixed with water and stirred to a thick paste, which is allowed to react with slightly more than the stoichiometric quantity of solid NaOH, which is added in small portions with constant stirring at high heat. As the reaction proceeds, the paste becomes increasingly liquid and stirring is easier. The reaction is brought to completion by further heating and stirring for one hour. The hydroxides are transferred to a 30- to 40liter earthenware container and stirred with 20-30 liters of water. The precipitate is allowed to settle, the supernatant is siphoned off, and the washing is repeated until the liquid gives only a weak alkaline reaction. The hydroxides are dissolved in conc. nitric acid (add 3%  $H_2O_2$  if large amounts of Ce are present). This nitrate solution serves as starting material for further workup.

Further separation of the resultant mixture of rare earths and isolation of individual components are described in later procedures in this section; in particular, see the preparation of pure La, Pr and Nd compounds from ceria earths by ion exchange and the preparation of pure Ce, Sm, Eu and Yb compounds.

**REFERENCES:** 

D. W. Pearce, R. A. Hanson, J. C. Butler, W. C. Johnson and W. O. Haas in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 38; S. J. Levy. The Rare Earths, London, 1924, pp. 18, 71, 79.

## Treatment of Gadolinite

Gadolinite is an yttria earth-beryllium-iron (II) silicate; its approximate composition is  $Y_2Be_2FeSi_2O_{10}$ , and it contains up to 50% rare earth oxides and about 10% BeO.

A) EXTRACTION, SEPARATION OF SILICIC ACID

The gadolinite is pulverized in a ball mill to a size passing through a U.S. 140-mesh sieve. Two kilograms of this material are mixed with seven liters of conc. HCl and evaporated (stirring) in a shallow porcelain dish placed on a sand bath; the evaporation is continued until a stiff paste is obtained (1.5 days required). The residue while still hot is taken up in 2-3 liters of hot water, suction-filtered and pressed dry.

To complete the extraction, the residue is again treated as above with 0.5 liter of conc. hydrochloric acid, taken up in one liter of water, suction-filtered, and washed with 0.5 liter of hot, dilute hydrochloric acid. The residue (SiO<sub>2</sub> and impurities not attached by hydrochloric acid) is discarded; it contains only about 0.5% of the rare earths present in the starting material.

#### B) PRECIPITATION WITH OXALIC ACID

The combined filtrates from (A) are diluted with water to twice their volume and heated to  $60^{\circ}$ C; a thin stream of a hot solution of 1400-1500 g. of oxalic acid dihydrate in two liters of water is then added with stirring. The mixture is kept warm overnight in a covered container; the precipitate is suction-filtered while still warm and washed with warm water. This yields about 95% of the rare earths initially present. Large quantities of rare earth oxalates are best converted to other compounds by calcining in a stream of air at 600-700 °C. The material is spread in a thin layer in a large crucible furnace and heated until the transient gray color disappears again. The oxides can then be readily dissolved in acids. They contain no more than about 0.1% Fe<sub>2</sub>O<sub>3</sub> and 0.15% SiO<sub>2</sub>.

# C) PRECIPITATION WITH NH<sub>3</sub>

The iron concentrates in the filtrate of the oxalate precipitation so that it constitutes more than 50% of the total metals present. To separate the Be and the rare earths, the latter are precipitated with NH<sub>3</sub> as the hydroxides (after transient reduction of the iron to the divalent state). The filtrate of operation (B) (two liters) is diluted with six liters of water in a 10-liter flask. The flask is provided with a dropping funnel, as well as gas inlet and outlet tubes; the gas inlet tube reaches to the bottom of the flask and the outlet terminates at the neck. The solution is heated to 60°C and the necessary amount of KMnO<sub>4</sub> [determined by testing a very dilute sample of the solution with Mn (II)] is added with efficient mixing to oxidize the oxalic acid. The highest possible concentration of KMnO<sub>4</sub> should be used. Then 500-700 g. of solid NH<sub>4</sub>Cl is added and the mixture is brought to a boil while SO<sub>2</sub> is passed through. The solution is maintained at the boiling point for an additional 30 minutes, a slow stream of SO<sub>2</sub> being bubbled through all the time. The flame is then removed, the SO<sub>2</sub> is replaced by a stream of H<sub>2</sub>, and conc. carbonate-free ammonia is added until no further precipitation occurs. The precipitate should be pure white. A brownish color indicates incomplete reduction of the Fe; a greenish tinge [impure Fe(OH)<sub>2</sub>] indicates that the amount of NH<sub>4</sub>Cl added was insufficient. In either case, the addition of ammonia is discontinued, the precipitate already formed is redissolved in hydrochloric acid, and the procedural fault is corrected. If the precipitate assumes a greenish color only toward the end of the operation, this is an indication that Fe(OH)<sub>2</sub> has begun to deposit because an excess of NH<sub>3</sub> is present in solution.

After completion of the precipitation the mixture is allowed to cool to  $45 \,^{\circ}$ C, and a freshly prepared solution of 70 g. of Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub> in 500 ml. of water and 20 ml. of dilute ammonia is added with efficient mixing. The passage of H<sub>2</sub> is discontinued and the precipitate is suction-filtered and washed with a warm (maximum  $45 \,^{\circ}$ C) solution of 10 g. of Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub> and 20 g. of NH<sub>4</sub>Cl in one liter of water, followed by one liter of pure water. The filtrate and wash water are discarded. The rare earths which have remained in solution after the oxalate precipitation, as well as the Be, are thus quantitatively precipitated. A solution in which the weight ratio BeO:Ln<sub>2</sub>O<sub>3</sub>:Fe<sub>2</sub>O<sub>3</sub> = 1:0.2:1.6 yields a precipitate in which this ratio is 1:0.2:0.05-0.03.

## D) SEPARATION AND RECOVERY OF THE Be

The moist hydroxide precipitate is dissolved in the minimum amount of warm glacial acetic acid and evaporated to complete dryness on a sand bath. Basic beryllium acetate  $Be_4O(CH_3COO)s$  is distilled from the residue at atmospheric pressure (m.p. 284°C; b.p. 330°C; cf. the procedure for the preparation of basic beryllium acetate, p. 901). The residue from this Be separation may be purified by another gadolinite extraction.

REFERENCE:

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# Pure La, Pr and Nd Compounds from Cerium Earths by Ion Exchange

Rare earth mixtures are efficiently separated by elution of complexes. Among the complexing agents which are usable as eluents, these which act as chelating agents possess significant advantages. A two-column process is usually employed: the first column is charged with the rare earth mixture and the second column with a suitable auxiliary cation. If the chelating agent used is readily soluble in water (for example, ethanolamine diacetic acid) the material in the second (or bottom) column may be in the H form.

The following process is suitable for rapid laboratory-scale separation of cerium earths: the cerium is removed from the mixture by precipitation with KMnO<sub>4</sub> and Na<sub>2</sub>CO<sub>3</sub> (see below, preparation of pure Ce compounds). The remaining compounds are adsorbed on 250 ml. of a cation exchange resin (Dowex 50 or Wofatit KPS-200; particle size 0.2-0.4 mm.). The resin is then placed in an ion-exchange column (I.D. 4 cm.) which is partly filled with water. A second column of the same I.D. is filled in the same manner with 350 ml. of the resin in the Zn form.

The eluent flows successively through the two columns; it contains 2% of nitrilotriacetic acid and is buffered with  $NH_3$  to a pH of 7.0. If the eluate flow rate is not less than 0.5 ml. per minute, there is no danger of formation of precipitates inside the columns. The eluate is collected in fractions; the lanthanides appear in the order of increasing ionic radius. The La, which remains in the columns after elution of the heavier earths, may itself be rapidly eluted with a solution containing 4% nitrilotriacetic acid and 2.4%  $NH_4C1$  (pH 9).

The eluate fractions are brought to a boil and the rare earths precipitated with oxalic acid. The mixtures are allowed to stand for 20 minutes at 80°C and filtered hot, and the solids are calcined to the oxides. The first fractions, which may contain minute quantities of Zn, are reprecipitated. More than 70% of the nitrilotriacetic acid used in the process can be recovered from the eluate by precipitation with HCl. Starting from 63 g. of the mixed oxide  $(14.7\% \text{ La}_2\text{O}_3, 23.5\% \text{ Pr}_8\text{O}_{11}, 54\% \text{ Nd}_2\text{O}_3, 5.4\% \text{ Sm}_2\text{O}_3, 2.4\% \text{ heavier earths}), this method gave 8.3 g. of La_2O_3 (yield 90\%, purity > 99.5\%), 10.3 g. of Pr_8O_{11} (yield 70\%, purity > 99\%) and 30 g. of Nd_2O_3 (yield 88.5\%, purity > 99\%) in 50 hours of elution.$ 

The method is also applicable to the yttrium earths, although in this case it requires more preliminary effort and takes longer.

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L. Wolf and J. Massonne. Chem. Techn. <u>10</u>, 290 (1958); L. Wolf and J. Massonne, J. prakt. Chem. (4) <u>275</u>, 178 (1956); L. Holleck and L. Hartinger. Angew. Chem. <u>66</u>, 586 (1954); <u>68</u>, 411 (1956); F. H. Spedding and J. E. Powell in: F. C. Nachod and J. Schuber, Ion Exchange Technology, New York, 1956, p. 365; J. Loriers and J. Quesney. Comptes Rendus Hebd. Séances Acad. Sci. <u>239</u>, 1643 (1954); J. Loriers. Ibid. <u>240</u>, 1537 (1955).

# Pure Cerium Compounds

# I. PRECIPITATION WITH Na2CO3 AND KMnO4

A nitric acid solution of 250 g. of cerium earths, which should consist only of nitrates, is diluted to one liter and brought to a boil, and the pH is adjusted to 2-3 with aqueous  $Na_2CO_3$ . This already precipitates some of the cerium in the form of Ce (IV) hydroxide. A solution of KMnO<sub>4</sub> is then added until the red color persists, the mixture is reheated to the boiling point, and the cerium is precipitated by adding (with constant stirring) a solution of KMnO<sub>4</sub> and  $Na_2CO_3$  (mole ratio 1:4). The pH gradually reaches 4 and the red color becomes more intense; stirring at the boiling point is then continued for an additional 10 minutes and the precipitate is suctionfiltered and washed with hot water. The filtrate contains minute amounts of cerium, in addition to the other rare earths. If it is desired to isolate the last traces of Ce, the pH of the boiling filtrate is adjusted to 5-6 with aqueous  $Na_2CO_3$ . The resultant precipitate consists of carbonates containing all the Ce present.

The cerium is isolated from the residues [which consist of Ce (IV) and Mn (IV) hydroxides] by solution of the residues in conc. hydrochloric acid and precipitation in the form of the oxalate. The precipitate is calcined to  $CeO_2$  (96% yield, 99.5% pure).

# II. PRECIPITATION WITH CaCO<sub>3</sub> AND KBrO<sub>3</sub>

The following procedure is suitable for a raw material mixture containing 40-50% CeO<sub>2</sub>: a nitric acid solution containing about 4100 g. of the rare earth oxides (total of 12 liters) is divided in

three equal portions. The solutions are heated in procelain dishes and adjusted to pH 2.7 by adding  $CaCO_3$  (mechanical stirring). Then 100 g. of KBrO<sub>3</sub> is added and the solutions are concentrated to one liter (see note below). This procedure is repeated several times to achieve complete oxidation and hydrolysis. If bromine vapor escapes during the evaporation, the pH is readjusted to 2.7 by adding further CaCO<sub>3</sub>. The mixtures are then diluted to five liters, heated almost to the boiling point, and allowed to settle overnight.

The supernatant liquors are decanted, combined and, after addition of 60 g. of  $KBrO_3$ , evaporated to 2-3 liters. The mixture is then diluted to eight liters, brought to a boil, and again decanted.

The combined precipitates are boiled in six liters of water, the supernatant liquor is decanted, and the precipitate is suctionfiltered through a Büchner funnel. The mother liquor is combined with the solution containing the other rare earths.

Other cerium compounds, for example, the basic nitrates, may be purified and converted to the oxides by boiling 50 g. of the moist starting material with 200 ml. of  $3N Na_2CO_3$  solution. The basic carbonate is filtered off and washed with 50 ml. of water. The product is dissolved in 16 N HNO<sub>3</sub> containing 3% H<sub>2</sub>O<sub>2</sub>.

This method gives 99.8% pure CeO<sub>2</sub> in 97.6% yield.

## NOTE:

The Ce (III) seems to oxidize during the evaporation in the hot conc. solution, while the hydrolysis of the resultant Ce (IV) takes place in the hot dilute solution.

# III. ETHER EXTRACTION OF Ce (IV) NITRATE

Very pure CeO<sub>2</sub> may be prepared from the commercial raw material by the following combined procedure: ten parts of crude oxide and seven parts of hydroquinone (reducing agent) are dissolved in boiling conc. hydrochloric acid. The hydroquinone oxidation products are destroyed by oxidative degradation with  $H_2O_2$  in ammoniacal solution, followed by evaporation with conc. nitric acid. Since it is accompanied by foaming, this operation must be carried out in a large vessel. If necessary the cerium may also be freed of organic contaminants by precipitation with oxalic acid, followed by solution of the precipitate in conc. nitric acid.

If the starting material also contains thorium, the oxalate is boiled with a concentrated, neutral to slightly ammoniacal, solution of ammonium oxalate; the precipitate is suction-filtered and thoroughly washed. Repetition of this operation yields a solution of  $(NH_4)_4Th(C_2O_4)_4$  which contains all the thorium initially present.

The nitric acid solution of nitrates, which is obtained in either case, is evaporated to dryness; it is then treated with  $NH_4NO_3$ 

(same weight as the initial  $CeO_2$ ) and evaporated several times with conc. HNO<sub>3</sub> until orange-red (NH<sub>4</sub>)<sub>2</sub>Ce(NO<sub>3</sub>)<sub>6</sub> begins to precipitate from the deep red solution. The precipitate is filtered off on a fritted glass filter (additional double nitrate may be recovered from the filtrate by further evaporation). The product is recrystallized from conc. nitric acid, dissolved in 6N nitric acid (free of nitric oxide), and extracted with peroxide-free ether (nitric oxides and peroxides reduce Ce<sup>4+</sup>). The residual material in the aqueous solutions should be reoxidized by evaporation with conc. nitric acid and extracted with ether, since the solutions still contain appreciable quantities of Ce<sup>3+</sup>.

The combined ether extracts are distilled, water being added during distillation. The cerium nitrate may be reprecipitated with water containing a hydrazine salt, which serves as a reducing agent.

The resultant cerium nitrate solution, which is about 2N in  $HNO_3$ , is filtered and slowly added in drops to a hot, concentrated solution of oxalic acid. The finely crystalline precipitate of cerium oxalate is suction-filtered, washed with a large quantity of water, dried and calcined to the oxide.

The oxide may also be obtained by evaporation of the nitrate solution, followed by thermal decomposition of the cerium nitrate.

Preparation of cerium compounds of especially high purity (for neutron bombardment): Peppard et al. recommend extraction of the Ce (IV) from a 10M HNO<sub>3</sub> solution with a 0.75M or 0.30M solution of bis(2-ethyl)hexyl orthophosphate in n-heptane [D. F. Peppard, G. W. Mason and S. W. Moline, J. Inorg. Nuclear Chem. 5, 141 (1957)].

## PROPERTIES:

Cerium (IV) oxide is white with a slight yellow tinge; the color is a function of particle size. Even slight contamination with Pr or Tb (0.005%) produces a distinct pink color; higher amounts cause a red-brown color. The calcined material is soluble in acids only in the presence of reducing agents. d (x-ray) 7.172. Crystal structure: type C1 (CaF<sub>2</sub> type).

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- I. E. J. Roberts. Z. anorg. allg. Chem. <u>71</u>, 305 (1911); J. Massonne. Private communication.
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(1950); R. Bock and E. Bock. Z. anorg. Chem. <u>263</u>, 146 (1950); U. Holtschmidt. Thesis, Freiburg i. Br., 1952; B. D. Blaustein and J. W. Gryder. J. Amer. Chem. Soc. <u>79</u>, 540 (1967).

#### Pure Samarium Compounds

#### I. REDUCTION WITH CALCIUM AMALGAM

 $2 \operatorname{SmCl}_{3} + \operatorname{Ca}(\operatorname{Hg}) = 2 \operatorname{SmCl}_{2} + \operatorname{CaCl}_{2}$ 513.6 40.1 442.7 111.0

A solution of 180-240 g. of anhydrous rare earth chlorides in 600-700 ml. of absolute ethanol is placed in a thick-wall, rubberstoppered separatory funnel. After addition of 7-10 g. of Ca in the form of a 1% amalgam (see p. 1804 for preparation), the separatory funnel is stoppered and vigorously shaken. Since this reduction to Sm (II) is accompanied by formation of calcium ethoxide and consequent evolution of H<sub>2</sub>, the flask must be frequently vented by opening the stopcock (without, however, allowing air to enter). The initially yellow solution soon becomes dark; after a few minutes the color becomes dark brown-red and precipitation of SmCl<sub>2</sub> begins. The Ca becomes exhausted after 20 minutes. The funnel is inverted so that it rests on the rubber stopper, a Bunsen valve is attached to the outlet tube, and the stopcock is opened. After 10 minutes, the CaO present in the mixture is neutralized by adding (through the funnel stem) 2-3 ml. of HCl-saturated anhydrous ethanol, and the funnel is vigorously shaken. The precipitate should turn bright red. After 30 minutes the Hg is separated and the finely crystalline precipitate of SmCl<sub>2</sub> is centrifuged in the absence of air. The mother liquor is decanted and the precipitate is freed of the adhering solution by shaking with air-free absolute ethanol, followed by centrifugation.

Further purification is achieved by taking up the precipitate in water, in which it is oxidized to Sm (III) and forms the basic chloride. Dilute hydrochloric acid is added and the mixture is heated on a water bath until the mercury left in the SmCl<sub>2</sub> has agglomerated and can be filtered off. The yellow solution is concentrated until crystallization just starts, and then saturated with HCl while cooling in ice. The precipitated hexahydrate is dehydrated and again reduced. The SmCl<sub>2</sub> obtained after this last purification procedure contains only a few percent of Eu.

The Eu may be removed by electrolysis of an alkaline acetate solution of the Sm-Eu mixture in the presence of lithium nitrate; a mercury cathode is used. Onstott, starting from a precipitate containing  $1.6\% \text{ Eu}_2O_3$ , was able to obtained a preparation entirely free of Eu in one electrolysis run.

K. WETZEL

II. REDUCTION WITH Mg + HCl

Another process suitable for the separation of Sm from rare earth mixtures consists in reducing the samarium, in the form of hydrated chlorides in ethanol or ethanol-dioxane, by means of Mg + HCl. A mixture containing 3% Sm can be concentrated to 55%Sm in a simple run.

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## Pure Europium Compounds

#### I. $EuCl_2 \cdot 2H_2O$

A conc. solution (d 1.35, 100 ml.) of rare earth chlorides containing about 70% Eu (balance Nd, Sm and Gd) is placed together with a few milliliters of conc. hydrochloric acid in a one-liter wide-neck flask. Zinc amalgam granules (100 g., U. S. standard mesh 80) are added, and the flask is stoppered and vigorously shaken by hand. From time to time it is held in front of the slit of a spectrometer to observe the absorption bands. The initially almost colorless solution turns yellowish; after about 30-40 minutes the Eu 5253 Å band disappears, indicating complete reduction to Eu (II). The solution is decanted from the remaining Zn while protected by a CO<sub>2</sub> blanket and poured into a second one-liter flask; the flask is closed with a two-hole stopper carrying a 250-ml. dropping funnel and a gas outlet capillary. Crystallization of EuCl<sub>2</sub> · 2 H<sub>2</sub>O starts after addition of the first 200 ml. of conc. hydrochloric acid; it reaches a maximum rate after 500 ml. has been added. This procedure precipitates 90% of the Eu present. After 2-3 hours of standing, the well-cooled mixture consists of almost equal volumes of pure white crystals with a faintly blue fluorescence and an essentially colorless mother liquor.

The air-sensitive crystals are filtered under  $CO_2$  through a cotton wool filter and suction-dried as far as possible. If oxidation does occur, the filter cake becomes hot and evolves HCl gas. The Eu may be further purified by redissolving the product chloride under  $CO_2$ . The operation should be repeated five times. Finally the product is filtered through a fritted glass filter (under a  $CO_2$  blanket) and washed with 10% methanolic HCl. The alcohol and the HCl can then be removed by slight heating in a fast  $CO_2$  stream.

1136

Another method for removal of traces of other earths present in crude europium consists in reduction of the europium with Zn amalgam in HCl solution. Then the trivalent earths are precipitated with carbonate-free ammonia. The Eu (II) remains in solution.

II. EuSO<sub>4</sub>

2 EuCl <sub>3</sub>	+ Zn =	$2  Eu Cl_2$	+	${\rm ZnCl}_2$
516.7	65.4	445.8		136.3
$EuCl_2$ +	$H_2SO_4 =$	= EuSO <sub>4</sub>	+	2 HCl
222.9	98.1	248.1		72,9

A solution of 3.5 g of  $Eu_2O_3$  in 5.4 ml of 6N HCl is diluted to about 200 ml. A Jones reductor (height 40 cm., diameter 2 cm.) is filled with 1% Zn amalgam granules (0.5-1 mm.), which are then washed with 200 ml. of 0.1N HCl. When the wash liquor just covers the zinc, the outlet of the reductor is dipped in 50 ml. of 8M  $H_2SO_4$ in a 600-ml. beaker covered with a round piece of paper. Carbon dioxide is then introduced into the beaker to expel the air. The EuCl<sub>3</sub> is passed slowly (2 ml./min.) through the reductor, followed by 150 ml. of 0.1N HCl. Very light, white, hairlike crystals of  $\alpha$ -EuSO<sub>4</sub> are the first product. This mixture is heated to 80°C in a CO<sub>2</sub> atmosphere, resulting in conversion of the  $\alpha$ -form to the stable  $\beta$ -form, which is less soluble in dilute H<sub>2</sub>SO<sub>4</sub> and settles as a dense crystalline mass. The mixture is cooled to room temperature, and the white EuSO4 is filtered and washed with dilute hydrochloric acid, followed by a few milliliters of HCl-acidified methanol. The CO<sub>2</sub> blanket is not necessary during the filtration. The product may be dried in air at 75°C. The yield is 90% of 99.7% EuSO4.

PROPERTIES:

White, microcrystalline. Sparingly soluble in water; d ( $25^{\circ}$ C) 4.98. Isomorphous with SrSO<sub>4</sub> and BaSO<sub>4</sub>.

III. EuCO<sub>3</sub>

$$EuSO_4 + Na_2CO_3 = EuCO_3 + Na_2SO_4$$
  
248.1 106.0 212.0 142.1

First, 5 g. of EuSO<sub>4</sub> is gradually added to 300 ml. of a vigorously boiling, oxygen-free solution which is 1N in Na<sub>2</sub>CO<sub>3</sub> and 0.4N in NaOH (12.6 g. of NaHCO<sub>3</sub> and 10.8 g. of NaOH). After a short time, the solution turns dark; the color disappears on further boiling, and a dense, lemon yellow, crystalline precipitate of EuCO<sub>3</sub> is formed. This precipitate is filtered and dried in air at 75 °C. An almost 100% pure product is obtained in 90% yield.

When larger quantities of  $EuCO_3$  are needed, the first fraction of the sulfate ions liberated in this reaction is removed by decantation and further  $Na_2CO_3$ -NaOH solution is added to the residue.

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- II. R. A. Cooley, D. M. Yost and H. W. Stone in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London; 1946, p. 70; H. N. McCoy. J. Amer. Chem. Soc. <u>57</u>, 1756 (1935).
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# Pure Ytterbium Compounds

## I. ISOLATION OF YTTERBIUM FROM YTTERBIUM EARTHS

IN THE FORM OF YbSO4

Reduction of  $Yb_2(SO_4)_3$  on a mercury cathode yields  $YbSO_4$ . The method is particularly suitable for the preparation of pure Yb from ytterbium earth mixtures.

The crude oxide, which must be free of foreign metals [which decrease the overvoltage necessary for the reduction of Yb (III) because they tend to form amalgams], is converted to the sulfate by evaporation with  $H_2SO_4$ . The electrolyte should contain 120 g. of sulfate and 50 g. of conc.  $H_2SO_4$  per liter. The electrolysis is carried out in a thick-wall beaker b (Fig. 285) with its bottom covered with a 1-cm. layer of very pure mercury. A nickel bus bar k connects the mercury pool to the negative side of the power supply. A carbon rod a, partially immersed in a clay cell c filled with dilute  $H_2SO_4$ , serves as the anode. A stirrer r, which agitates both the mercury surface and the electrolyte, prevents the formation of a dense precipitate on the cathode and thus makes possible the preparation of larger quantities of YbSO<sub>4</sub>. The electrolyte temperature is maintained at 20°C by external cooling with running water.

The electrolysis is carried out at 72 volts and a current density of  $0.05 \text{ amp./cm}^2$  (about 4-4.5 amp. if the beaker is 10 cm. in diameter). At higher current densities the formation of crystals of YbSO<sub>4</sub> is so rapid that they occlude considerable quantities of impurity ions (Tm, Lu, etc.) After a few minutes the

solution turns green. When the cathode becomes covered with a loose layer of YbSO<sub>4</sub> 2-3 cm. thick, the current efficiency becomes very low and the electrolysis is stopped. Under the conditions described, the process requires 2-3 hours.

The precipitate is collected on a Büchner funnel and washed with water, and the residual water is rapidly removed by suction. Speed is necessary because the oxidation of YbSO<sub>4</sub> is accompanied by a marked temperature increase, which could cause decomposition of the product YbSO<sub>4</sub> [or Yb<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub>].

The precipitate is dissolved in dilute nitric acid, neutralized with ammonia, and reprecipitated with oxalic acid. The oxide obtained upon calcination of the oxalate still contains traces of the sulfate.

Additional quantities of YbSO<sub>4</sub>



Fig. 285. Electrolytic preparation of ytterbium (II) sulfate. *a* carbon rod, *b* beaker, *c* clay cell.

may be recovered from the spent electrolyte by inclusion in the isomorphous SrSO<sub>4</sub>. If this is desired, then the electrolyte should contain only 0.5% H<sub>2</sub>SO<sub>4</sub>. The SrSO<sub>4</sub> solution is prepared by very rapid neutralization of 3 g. of SrCO<sub>3</sub> with the stoichiometric quantity of dilute H<sub>2</sub>SO<sub>4</sub>. This solution is added one hour after the start of the electrolysis. The added solution contains slowly crystallizing SrSO<sub>4</sub>. The addition is repeated twice at intervals of one hour. After 4-5 hours the precipitate, which contains SrSO4 and YbSO<sub>4</sub> in a ratio of about 10:1, is filtered off and washed. The YbSO<sub>4</sub> is protected against oxidation to Yb (III) by inclusion in the SrSO<sub>4</sub> lattice. On calcination in air; YbSO<sub>4</sub> is converted to Yb<sub>2</sub>O<sub>3</sub> and may be separated from the SrSO<sub>4</sub> by digestion with conc. hydrochloric acid on a water bath. Some additional SrSO<sub>4</sub> may be precipitated from this HCl solution by adding dilute sulfuric acid and allowing the solution to stand 12 hours. After removal of the SrSO<sub>4</sub>, the Yb is precipitated with oxalic acid in the usual manner.

The electrolytic separation of Yb is accompanied by concentration of Tm and Lu in the residual solution.

Europium and samarium may be separated (as  $EuSO_4$  and  $SmSO_4$ ) from the other rare earth metals by the same method.

PROPERTIES:

Formula weight 269.11. Green  $Yb^{2+}$  ions are oxidized by water to  $Yb^{3+}$  (evolution of H<sub>2</sub>). Solubility in dilute sulfuric acid:

4 g. YbSO<sub>4</sub>/liter of 1% H<sub>2</sub>SO<sub>4</sub> (0.2N); 8 g. YbSO<sub>4</sub>/liter of 5% H<sub>2</sub>SO<sub>4</sub> (1N); 20 g. YbSO<sub>4</sub>/liter of 12.5% H<sub>2</sub>SO<sub>4</sub> (2.5N). Isomorphous with SrSO<sub>4</sub>.

# II. PURIFICATION OF Yb (Sm, Eu) VIA AN AMALGAM PROCESS

 $Yb^{3+} + 3 Na (Hg) = Yb (Hg) + 3 Na^{+}$ 

The procedure is suited both for efficient purification of a concentrated ytterbium solution (method a below) and for isolation of Yb from a mixture of neighboring rare earths, as well as freeing the latter from Yb (method b).

a) A product containing about 97% Yb<sub>2</sub>O<sub>3</sub> (balance is ytterbium earths), which may be prepared via YbSO<sub>4</sub> by the method described under I. is dissolved in acetic acid on a water bath and evaporated until crystallization. A solution of 107 g. of ytterbium acetate in 133 ml. of boiling water is prepared in a one-liter flask. The hot solution is vigorously shaken for two minutes with 250 ml. of 0.5% sodium amalgam (125% of the theoretical Na). During the reaction 3 ml. of acetic acid is added to prevent the formation of hydroxide (formation of NaOH by partial decomposition of the sodium amalgam). It is best not to shake the mixture until the Na is fully spent, since shaking may cause the Yb to partially redissolve in the form of Yb (II) ions (green color of the aqueous layer). In addition, the Yb content of the amalgam should not exceed 1% to avoid solidification. The amalgam is separated from the solution and water-washed twice to remove the acetate. It is then treated with sufficient dilute hydrochloric acid to neutralize the residual Na. Small quantities of Yb which go into solution during this step are precipitated with NaOH. The amalgam is then shaken with excess hydrochloric acid until calomel starts to form. The Yb(OH), precipitate is added to the HCl solution and the mixture is evaporated to a sirup. The precipitating NaCl is filtered off. The filtrate is calcined. Spectroscopically pure Yb<sub>2</sub>O<sub>3</sub> is obtained in a yield exceeding 90%.

The acetate solution remaining from the first extraction may be allowed to react, after addition of 3 ml. of acetic acid, with an additional 83 ml. of sodium amalgam. The resultant ytterbium amalgam is worked up as above. The mother liquor is converted to the hydroxide and may be reextracted after dissolving in acetic acid.

b) If complete extraction of Yb from a mixture of ytterbium earths is desired, the solution must be shaken up to 20 times with sodium amalgam and the aqueous layer repeatedly freed of the sodium acetate formed, since high concentrations of the latter interfere with the reaction. Using this method, Marsh was able to reduce the Yb content of a Lu preparation to 0.0033%. Pure compounds of Sm and Eu may be prepared by a basically similar method. The preferential formation of Sm, Eu and Yb amalgams is due to the fact that metallic Eu and Yb always form divalent ions, while Sm does so partially.

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## **Metallic Rare Earths**

#### I. REDUCTION OF THE CHLORIDES WITH METALS

LANTHANUM METAL, POWDER

$$L_aCl_3 + 3K = L_a + 3KCl_{245.3} + 117.3 + 138.9 + 223.7$$

The Vycor (or similar glass) apparatus shown in Fig. 286 is dried by fanning with a flame under high vacuum. A vacuum pump is connected at d; c and i are closed off with rubber stoppers. The apparatus is filled with pure, dry nitrogen or argon, and a small tube containing distilled potassium is placed in tube b; the neck of the potassium ampoule is broken immediately prior to use. Anhydrous rare earth chloride is introduced into a through tube c(air must be excluded during this operation) and tube c is immediately closed off again. The apparatus is melt-sealed at c and i, and evacuated through d to a high vacuum. Capillary f is heated and bent downward (to position b') and the potassium is slowly distilled from b' into g. This second distillation removes the possibility of traces of potassium oxide coming into contact with the chloride. The apparatus is then melt-sealed at f and e, a small portion of the potassium is distilled into constriction h, and the tube (a-g) is heated to 220-350°C in an electric furnace. Part of the rare earth chloride is reduced after a short time; an additional fraction of the potassium is then distilled into h, and the tube is reheated in the furnace. This stepwise reduction is continued until most of the rare earth chloride has been converted to the metal. Only then is the remaining potassium distilled by turning the tube upside down and placing almost the entire apparatus (f-a)in the furnace. The potassium is then immediately distilled off


Fig. 286. Preparation of metallic rare earths by reduction of the chloride with metallic potassium. a rare earth chloride; b, b' metallic potassium; c filling tube for the rare earth chloride; d connection to vacuum pump.

again, and the process is repeated several times. This stepwise reduction prevents the reactants from fusing together, and thus ensures completion of the reaction. Finally the tube is gradually pulled out of the furnace, until a potassium mirror no longer forms at the unoccupied spots on the tube between g and h. The product consists of a loose black powder which does not adhere to the glass walls of the tube.

After cooling, tip f is connected via a dry rubber tube to the vacuum-N<sub>2</sub> (or Ar) system and broken off under N<sub>2</sub> or Ar. The alkali metal at g serves as a barrier and traps any traces of water vapor which may be introduced. The tube can now be broken at h without endangering the product, and the mixture of rare earth metal + 3 alkali chloride at a may be poured into a transfer apparatus through which protective gas is flowing (for transfer apparatus see Part I, p. 75 ff., especially Fig. 54).

All the rare earth metals, in the form of powders mixed with alkali chloride, may be prepared by this method. In preparing Sm, Eu or Yb metals (these elements form divalent compounds), a temperature of 250°C must not be exceeded, since at higher temperatures, the direction of the reaction is reversed and SmCl<sub>2</sub>, EuCl<sub>2</sub> and YbCl<sub>2</sub> are formed.

CERIUM METAL, SOLID

 $2 \operatorname{CeCl}_{3} + 3 \operatorname{Ca} = 2 \operatorname{Ce} + 3 \operatorname{CaCl}_{2}$ 493.0 120.3 280.3 333.0

A crucible of sintered CaO or dolomite is placed in an iron tube provided with a welded-on bottom and a screw lid, and the space between the crucible and the tube is filled with CaO powder. This precaution prevents contact between the reaction mixture and the iron wall if the crucible should break. Since the heat of the reaction between Ca and CeCl<sub>3</sub> is not sufficient for clean separation between the metal ingot and the slag, it is necessary to add a third component which produces a highly exothermic reaction with Ca, e.g., I<sub>2</sub>, S, KClO<sub>3</sub> or ZnCl<sub>2</sub>.

For a tube 20 cm. high and 2.5 cm. in diameter, suitable quantities of reactants are 200 g. of CeCl<sub>3</sub>, 103 g. of I<sub>2</sub> (mole ratio I:CeCl<sub>3</sub> = 0.625:1.0) and a 15% excess of very pure Ca powder (particles 0.3-2 mm.). The reactants are mixed under anhydrous condition and placed in the crucible; the iron cap is filled with CaO and screwed on. The tube is placed in a furnace heated to 650-750 °C. The reaction starts suddenly when the temperature inside the tube reaches 400 °C. The yield of Ce metal is 93%. The reaction may be carried out on a larger scale, but due to smaller relative heat losses, only 0.5 mole of I<sub>2</sub> per mole of CeCl<sub>3</sub> and a 10% excess of Ca are needed. The use of sulfur or KClO<sub>3</sub> lowers the yield. The resultant Ce metal contains 1-5% Ca and 0.1-1% Mg.

When smaller quantities of raw materials are used, the reaction temperature must be increased. This is done by replacing the iodine with  $ZnCl_2$  (3-6% Zn, based on the amount of Ce). The product is freed of zinc by evaporating it in vacuum. The yield is 90% of Ce containing only 0.002% Zn.

Any Ca, Mg or Zn which may be dissolved in the Ce is removed by placing the product in a crucible made of MgO, CaO, BeO or Ta, which in turn is placed in a second crucible made of graphite. This assembly is placed in a quartz tube with one end closed and the other connected to a high-vacuum pump via a watercooled brass coupling. The coupling is provided with a glass window to facilitate optical temperature measurement. The wellinsulated quartz tube is placed for 30 minutes in an induction furnace heated to 1250°C. The melt is held at this temperature for 10-15 minutes, until cessation of bubbling.

This entire sequence of procedures can be used to prepare La, Ce, Pr and Nd in 99.9% purity. The reaction with Ca converts  $SmCl_3$ ,  $EuCl_3$  and  $YbCl_3$  to the dichlorides. The preparation of Y fails due to the high melting point of this metal.

### II. REDUCTION OF THE OXIDES WITH METALS

### SAMARIUM METAL, SOLID

A tantalum crucible (height 20 cm., diameter 2.5 cm.) containing a mixture of 20 g. of  $Sm_2O_3$  and 20 g. of freshly prepared La turnings is heated for 30 minutes at 1450°C in an electric furnace at a pressure of 0.001 mm. The upper part of the crucible projects from the furnace and is closed off with a lid carrying a connection to a high-vacuum pump. The Sm is deposited on the cooler parts of the crucible. The method results in a highly pure metal free of La; the yield is 80%.

Pure Yb metal may be prepared by the same method. In a similar preparation the La may be replaced by Ca or Al.

PROPERTIES:

Atomic weight 150.35. Silvery, air-stable metal. More volatile than La metal.

REFERENCE:

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### III. ELECTROLYSIS OF FUSED CHLORIDES

### LANTHANUM METAL, SOLID

The apparatus for melt electrolysis is shown in Fig. 287. The anode is a graphite crucible (inside diameter 40 mm., inside height 80 mm., wall thickness 5 mm., bottom thickness 10 mm.) containing the melt. The current is supplied through a sleeve surrounding the bottom of the crucible; the sleeve is connected to the power supply through an iron rod. The cathode is a Mo rod (diameter 10 mm., length about 100 mm.) friction-fitted into an iron pipe and covered near the top with a tube of sintered corundum cemented in with a talc-waterglass mixture. The rotating cathode should be able to agitate the melt; the current is supplied via a carbon friction contact. To collect the La metal which is thrown off by the spinning cathode and protect it against contact with graphite and the  $Cl_2$  formed at the anode, a sintered alumina crucible (upper diameter 40 mm., height 30 mm.) is fitted exactly into the graphite crucible.

The furnace is heated to 1000 °C and the crucible is charged with a salt mixture of the following composition: 27.4% LaCl<sub>3</sub>, 68.0% KCl, 4.6% CaF<sub>2</sub> (3.75 g. of KCl and 0.25 g. of CaF<sub>2</sub> per gram of La<sub>2</sub>O<sub>3</sub>). It is advisable to add initially only a small portion of the fluxing material. The mixture is allowed to melt and any NH<sub>4</sub>Cl present is allowed to escape; the remainder of the flux is then added during the first minutes of the actual electrolysis. The electrolysis is run at 6-8 volts and 40 amp. The highest current efficiencies and product yields are obtained at about 7 amp. per cm.<sup>2</sup> and 25 amp.-hr. The cathode should rotate at a rate of 1-2 r.p.s. At the end of the reaction, the current is shut off and stirring (rotation of the cathode) is continued for a few minutes. The crucible is then removed, allowed to cool and broken to pieces. The resultant ingot contains more than 99% lanthanum.

With some of the other rare earth metals (Sc, Gd), it is necessary to work below the melting point of the metal. In such cases the metal is deposited on a cathode of molten Zn or Cd, in which the metal dissolves. The Zn or Cd is vacuum-distilled from the product alloy.

PROPERTIES:

Atomic weight 138.92. Iron gray, with a vivid metallic luster when polished; ductile, malleable. Tarnishes rapidly in moist air. M.p.  $885^{\circ}$ C; d<sup>20</sup> 6.18. Crystal structure:  $\alpha$ -La, type A3 (Mg type);  $\beta$ -La, type A1 (Cu type).

#### CERIUM METAL, SOLID

Metallic cerium is obtained via electrolysis of a fused mixture of anhydrous  $CeCl_3$  and KCl-NaCl.

The reaction is carried out in a roughly cylindrical copper vessel, with a wall thickness of 1 mm. (see Fig. 260, p. 957). The inside diameter is about 2.5 cm. At the top, the tube widens to an inverted truncated cone with a base diameter of about 8 cm. The cathode is a carbon rod (diameter 9 mm., length 16 cm.) inserted from below; up to about 1.5 cm. from the upper end of the cylindrical section of the tube, the cathode is wrapped with asbestos cord; this asbestos cord, in conjunction with the unmelted portion of the chloride mixture which rests on it, serves as the bottom of the melting pot. The anode is a somewhat thicker carbon rod inserted from above. The position of the anode may be adjusted by a heightregulating device attached to the side of this crucible. The electrode gap is located approximately at the midheight of the conical melting space.

For small-scale preparations, a thin carbon rod about 3 mm. in diameter and 20 mm. long is clamped between the two cathodes. The crucible is filled with 200 g. of CeCl<sub>3</sub> and 15-20 g. of KCl-NaCl (equimolar mixture) and the crucible contents are melted as rapidly





as possible with a current of 30-40 amp. at 12-15 volts. As soon as the melt thins in consistency, the anode is raised somewhat, the thin carbon rod is removed, and electrolysis is carried out for several hours at 700-750°C. After solidification the metallic ingot is removed and remelted under KCl-NaCl in a silicon carbide crucible.

PROPERTIES:

Atomic weight 140.13. Iron gray, with a vivid metallic luster when polished; may be cut with a knife; somewhat harder than lead; ductile, malleable. M.p. 635°C; d<sup>25</sup> 6.92. Tarnishes rapidly in moist air; burns at 160-180°C in a stream of O<sub>2</sub>. Attacked by water (evolution of H<sub>2</sub>). Crystal structure:  $\alpha$ -Ce, type A3 (Mg type);  $\beta$ -Ce, type A1 (Cu type).

# IV. ELECTROLYSIS OF ALCOHOLIC CHLORIDE SOLUTIONS

The electrolysis of an alcoholic solution of a rare earth chloride on a mercury cathode (20 volts, current density  $0.02 \text{ amp./cm}^2$ ) yields an amalgam with a rare earth metal concentration of up to 3%. The mercury is removed by vacuum distillation, leaving behind the pure rare earth metal.

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# Rare Earth Trichlorides

LnCI<sub>3</sub> (anhydrous)

I. REACTION OF THE OXIDES WITH Cl<sub>2</sub> AND S<sub>2</sub>Cl<sub>2</sub>

The rare earth oxide (1-2 g.) is placed in a porcelain boat and chlorinated for about five hours in a stream of  $Cl_2-S_2Cl_2$  (prepared

by bubbling  $Cl_2$  through a wash bottle filled with  $S_2Cl_2$  and standing in a 30-40 °C water bath). The temperature is slowly raised during the process from an initial 400 °C to about 20 ° below the melting point of the chloride. The chlorides deposit on the bottom as loose powders ready for use in further reactions.

The chlorides of Sm, Eu, Gd, Tb, Dy and Y, which melt below 700 °C, are best prepared by dehydration of the hydrated chlorides in a stream of HCl.

II. DEHYDRATION OF THE HYDRATED CHLORIDES IN A STREAM OF HCl

The hydrate of the rare earth chloride (3-5 g.) is dried in a vacuum desiccator and then heated by stages in the region of the individual hydrate transition temperature while a stream of oxygen-free N<sub>2</sub>-HCl mixture is passed over it. The boat, which may be of porcelain, quartz, gold or platinum, is placed in a tube of Pyrex, Vycor or quartz. The temperature may be raised beyond the transition region only when no further hydrochloric acid condenses on those sections of the tube which project from the furnace; if this precaution is not observed the chloride melts in the water of crystallization and the product then contains oxychlorides. Dehydration is complete after 30-60 hours. Heating at 300-400°C in a stream of pure HCl is continued for one hour, and the product is allowed to cool in a stream of N<sub>2</sub>. The stopcocks and ground joints which come into contact with the hydrogen halides are greased with a mixture of paraffin and paraffin oil.

The product must yield a clear solution in water. Contamination with traces of oxychloride may be recognized by turbidity of the aqueous solutions.

### III. HEATING OF THE HYDRATED CHLORIDES WITH NH<sub>4</sub>Cl

Dehydration of the hydrated chlorides may also be achieved by heating with an excess (1-1.5 times) of NH<sub>4</sub>Cl. The products, however, always contain small quantities of NH<sub>4</sub>Cl.

### IV. HEATING OF THE OXIDES WITH NH<sub>4</sub>Cl

A 250-ml. quartz Erlenmeyer flask equipped with an adapter that can be closed off with a small glass cap and can also be connected to a quartz tube (length 25 cm., diameter 3 cm.) is filled with a mixture of 60 g. of the rare earth oxide and 120 g. of  $NH_4Cl$ .

The flask, tilted about 30° from the horizontal, is slowly rotated around its axis and heated to 220-250°C on an air bath. Evolution of NH<sub>3</sub> ceases after 6-8 hours. After a short cooling

period, the flask is closed with the glass cap while still warm and then allowed to cool completely. The cap is then replaced with the quartz tube. The other end of the tube is connected (via two gas traps) to an oil vacuum pump. The flask is evacuated and surrounded with an electric furnace extending a few centimeters beyond the quartz joint. The mixture is slowly heated to 300-350°C, resulting in evolution of a small quantity of water vapor and NH<sub>3</sub>. The excess NH<sub>4</sub>Cl sublimes into the quartz tube. To prevent cracking of the glass connection, it is sometimes necessary to cool the other end of the tube with a water-cooled lead or tin coil. After 4-5 hours, the mixture is allowed to cool, the quartz tube is cleaned, and the sublimation is repeated. Complete removal of the last traces of NH<sub>4</sub>Cl is attained only at 400°C.

The method is particularly suited to the preparation of large quantities of product. The oil pump may be replaced by a jet ejector if an adequate trap filled with a drying agent is inserted in the line.

The above methods are suitable for the preparation of all the anhydrous rare earth chlorides, including that of yttrium. For the preparation of ScCl<sub>3</sub>, see W. Fischer, R. Gewehr and H. Wingchen, Z. anorg. allg. Chem. 242, 170 (1939).

### PROPERTIES:

Hygroscopic powders which give clear solutions in water and alcohol. The melting points drop from LaCl<sub>3</sub> (~860°C) to TbCl<sub>3</sub> (~600 °C) and rise again to LuCl<sub>3</sub> (~900 °C).

### REFERENCES:

- I. W. Klemm, K. Meisel and H. U. von Vogel. Z. anorg. allg. Chem. 190, 123 (1930).
- II. O. Hönigschmid and H. Holch. Z. anorg. allg. Chem. 165, 294 (1927); 177, 94 (1928); L. Holleck. Angew. Chem. 51, 243 (1938).
- III. F. Weibke and J. Sieber. Z. Elektrochem. <u>45</u>, 518 (1939). IV. A. Brukl. Angew. Chem. <u>52</u>, 152 (1939); J. B. Reed. J. Amer. Chem. Soc. 57, 1159 (1935); D. H. West and B. S. Hopkins. Ibid. 57, 2185 (1935).

### Rare Earth Tribromides

### LnBr<sub>a</sub> (anhydrous)

I. DEHYDRATION OF HYDRATED BROMIDE-NH<sub>4</sub>Br MIXTURES IN A STREAM OF HBr

A hydrobromic acid solution containing six moles of NH<sub>4</sub>Br per mole of the rare earth bromide is carefully evaporated to dryness on an air bath, with constant stirring toward the end of the operation. The evaporation should be carried out in a stream of  $CO_2$ . The crumbly salt mixture is dehydrated in a stream of HBr at slowly increasing temperatures. The product must not be allowed to melt under any circumstances! The sublimation of NH<sub>4</sub>Br starts at 250 °C; its last traces are removed at 600 °C.

Very pure tribromides are obtained by dehydration and removal of  $NH_4Br$  from the  $LnBr_3 \cdot 6 H_2O-NH_4Br$  mixture on heating in high vacuum. For the preparation of ScBr<sub>3</sub>, see W. Fischer, R. Gewehr and H. Wingchen, Z. anorg. allg. Chem. <u>242</u>, 170 (1939).

### II. TREATMENT OF THE ANHYDROUS CHLORIDES WITH HBr

The anhydrous rare earth chloride (1-2g.) is heated in a stream of HBr for about seven hours. The temperature is slowly raised from 400 °C to slightly below the melting point of the bromide.

PROPERTIES:

Hygroscopic powders. The melting points rise with atomic weight from  $SmBr_3$  (~628°C) to  $ErBr_3$  (~950°C).

**REFERENCES**:

- I. G. Jantsch et al. Z. anorg. allg. Chem. <u>207</u>, 361 (1932);
   G. Jantsch and N. Skalla. Ibid. <u>201</u>, 213 (1931).
- II. W. Klemm and J. Rockstroh. Z. anorg. allg. Chem. <u>176</u>, 189 (1928).

### Rare Earth Triiodides

### LnI<sub>3</sub> (anhydrous)

# I. DEHYDRATION OF HYDRATED IODIDE-NH<sub>4</sub>I MIXTURES IN A STREAM OF HI-H<sub>2</sub>

A mixture of one mole of  $LnI_3 \cdot 6 H_2O$  and six moles of  $NH_4I$  is heated in a stream of  $HI + H_2$  mixture with a moderate HI concentration. Under no circumstances should the temperature be raised at a rate fast enough to melt or sinter the product during the dehydration. Because the product is extremely sensitive to  $O_2$  and moisture, great care must be exercised in purifying the gases. Since the last traces of  $NH_4I$  sublime only at high temperatures, the mixture is heated to 600°C during the last stage. When the dehydration is complete, the HI is flushed out with  $N_2$ . The iodides are stored under  $N_2$ .

This method is suitable for the iodides of La, Pr and Nd. However,  $SmI_3$  is usually contaminated with some  $SmI_2$ , which is converted to  $SmI_3$  after elimination of the  $NH_4I$ . This is done by heating the product to 500 °C and treating it with iodine vapor. Again, under the conditions of this method,  $EuI_3$  decomposes to  $EuI_2$  and iodine. Because of their tendency to hydrolyze, the iodides of the rare earth metals which are less electropositive than Eu can be prepared only from the anhydrous chlorides. The same applies to  $CeI_3$ .

II. REACTION OF THE ANHYDROUS CHLORIDES WITH MIXTURES OF HI +  $\mathrm{H}_2$ 

The anhydrous rare earth chloride is heated for 4-6 hours until 600°C is reached; it is then held at this temperature for 30-40 hours in a stream of HI-H<sub>2</sub> containing as much HI as possible. The iodides are cooled and stored under N<sub>2</sub>.

Special care must be exercised with the lower-melting chlorides, since the chlorides no longer react with the HI when enveloped in iodide.

PROPERTIES:

Hygroscopic powders. The melting points drop from LaI<sub>3</sub> (~761°C) to  $PrI_3$  (~733°C) and rise again to LuI<sub>3</sub> (~1045°C).

**REFERENCES:** 

- I. G. Jantsch et al. Z. anorg. allg. Chem. <u>185</u>, 56 (1930);
   E. Hohmann and H. Bommer. Ibid. <u>248</u>, 384 (1941).
- II. G. Jantsch et al. Z. anorg. allg. Chem. 201, 207 (1932); 207, 353 (1932); 212, 65 (1933); E. Hohmann and H. Bommer. Ibid. 284, 383 (1941).

# **Rare Earth Dihalides**

LnX<sub>2</sub> (anhydrous)

The trihalides of Sm, Eu and Yb are converted to the dihalides by heating in a stream of carefully purified  $H_2$ . The temperature should not be raised as high as the melting point of the trihalide, since the molten compounds react either slowly or not at all with  $H_2$ .

All the dichlorides, dibromides and diiodides of Sm, Eu and Yb can be prepared by this method. Note that  $EuI_2$  is formed under the conditions given for the preparation of the triiodides (method II). Thermal degradation of YbI<sub>3</sub> in high vacuum is the preferable method for obtaining YbI<sub>2</sub>.

**REFERENCES:** 

W. Prandtl and H. Kögl. Z. anorg. allg. Chem. <u>172</u>, 265 (1928);
G. Jantsch, H. Rüpnig and W. Runge. Ibid. <u>161</u>, 210 (1927);
W. Kapfenberger. Ibid. <u>238</u>, 281 (1938); G. Jantsch, N. Skalla and H. Jawurek. Ibid. 201, 218 (1931).

### Cerium (III) Oxide

### $Ce_2O_3$

Reduction of  $CeO_2$  in a stream of  $H_2$  is the best method. It is carried out in a silicon carbide boat. The  $H_2$  must be carefully purified (free of oxygen) and dried. Very pure  $CeO_2$  (3 g.), prepared as described on p. 1133, requires about 80 hours at 1000°C (or 45 hours at 1100°C) for complete reduction. Traces of La and Nd moderately increase the rate of reduction, while Pr and Tb do so markedly. The reduction is complete when the blue-black color of the partially reduced intermediates changes to the pure golden yellow of  $Ce_2O_3$ .

PROPERTIES:

Golden yellow (greenish yellow products are incompletely reduced); converts slowly in air to  $CeO_2$ ; rapid conversion, accompanied by a glow, on slight heating. Readily soluble in acids. d (x-ray) 6.856. Crystal structure: type D5<sub>2</sub> (A-sesquioxide type).

**REFERENCES:** 

E. Friederich and L. Sittig. Z. anorg. allg. Chem. <u>134</u>, 316 (1925); <u>145</u>, 127 (1925); W. Zachariasen. Z. phys. Chem. <u>123</u>, 134 (1926); G. Brauer and U. Holtschmidt. Z. anorg. allg. Chem. <u>265</u>, 105 (1951); U. Holtschmidt. Thesis, Freiburgi. Br., 1952; G. Brauer and U. Holtschmidt. Z. anorg. allg. Chem. <u>279</u>, 129 (1955); D. J. M. Bevan. J. Inorg. Nuclear Chem. <u>1</u>, 49 (1955).

# Praseodymium (IV) Oxide

### PrO<sub>2</sub>

At 400 °C, praseodymium oxide preparations achieve a composition corresponding to  $PrO_2$  only under an  $O_2$  pressure of 282 atm. However, at 300 °C, only 50 atm. of  $O_2$  suffices. The oxidation of



Fig. 288. Quartz tube for oxidation of praseodymium oxide.  $\Pr_{0}O_{11}$  is carried out in the quartz tube shown in Fig. 288. The  $O_{2}$  is generated by the decomposition of a weighed amount of NaClO<sub>3</sub> in the sealed tube. The amount of chlorate used is calculated beforehand to yield an  $O_{2}$  pressure of 50 atm. in the sealed tube. After the NaClO<sub>3</sub> is decomposed by local heating, the entire tube is heated to 300°C for 48 hours.

PROPERTIES:

Dark brown powder, easily reduced by  $H_2$  to  $Pr_2O_3$ . Crystal structure: type C1 (CaF<sub>2</sub> type). Measurements of the dissociation pressure of  $O_2$  indicate that the Pr-O system contains additional stable phases with compositions of  $PrO_{1.715}$ ,  $PrO_{1.602}$  and  $PrO_{1.833}$ .

REFERENCES:

J. D. McCullough. J. Amer. Chem. Soc. <u>72</u>, 1386 (1950); W. Simon and L. Eyring. Ibid. <u>76</u>, 5872 (1954); R. E. Ferguson, E. D. Guth and L. Eyring. Ibid. <u>76</u>, 3890 (1954).

### Rare Earth Hydroxides

# Ln(OH)<sub>3</sub> (crystalline)

Crystalline trihydroxides  $Ln(OH)_3$  of the lanthanides (at least those ranging from La to Er) and of Y are prepared by heating the hydroxides under conc. (7N) NaOH: A solution of 2 g. of the nitrate in 2 ml. of water is added to a silver crucible containing a solution of 7 g. of NaOH in 7 ml. of water. The crucible, covered with a silver lid, fits precisely into a pressure tube closed off with a screwed-on cap. The tube is heated for 25 hours at 200°C. The mixture is cooled, the clear supernatant is siphoned off, and the product is washed several times (by decantation) with  $CO_2$ -free water; it is then washed on a filter crucible in the absence of  $CO_2$ and dried by suction. Final drying is achieved by storing the product for 24 hours over conc.  $H_2SO_4$  in a vacuum desiccator.

PROPERTIES:

Transparent, hexagonal prisms. Solubility in conc. NaOH increases with the atomic number. The dehydration passes through an intermediate stage, MO(OH), in which the compounds have the PbFCl structure. La $(OH)_3$  has UCl<sub>3</sub> structure.

REFERENCES:

R. Fricke and A. Seitz. Z. anorg. allg. Chem. <u>254</u>, 107 (1947); R. Roy and H. A. McKinstry. Acta Crystallogr. (Copenhagen) <u>6</u>, 365 (1953).

### Lanthanum Sulfide

### La<sub>2</sub>S<sub>3</sub>

Anhydrous LaCl<sub>3</sub> is heated in a stream of pure  $H_2S$ . The temperature is maintained at 500-600°C for several hours, followed by heating at 600-700°C overnight. Prior to use, the  $H_2S$  is dried over CaCl<sub>2</sub> and  $P_2O_5$ , liquefied at -78°C (see p. 344 ff.) and evaporated from the liquid [A. Simon, Ber. dtsch. chem. Ges. <u>60</u>, 568 (1927) and this Handbook, Part I, p. 46 ff.] at a flow rate of one bubble per second. The intermediate product is then heated to 800-1000°C for several hours and allowed to cool in a stream of  $H_2S$ .

This method is suitable for all the rare earth sulfides, including those of Sc and Y.

$$\begin{array}{rrrr} La_2(SO_4)_3 &+ & 12 H_2S &= La_2S_3 &+ & 12 S &+ & 12 H_2O \\ 566.0 & & 409.0 & & 374.0 & & 384.8 & & 216.2 \end{array}$$

The recrystallized sulfate hydrates may also serve as starting materials. Except for the fact that the decomposition temperatures of the dehydrated sulfates (given by Brill; see references below) are different, the procedure is similar to that given above for the chlorides. In this case, however, the products are usually contaminated with variable amounts of  $Ln_2O_2S$ .

Under these conditions  $Y_2(SO_4)_3$  and  $Er_2(SO_4)_3$  form  $Ln_2O_2S$  exclusively. It is also possible to prepare  $La_2O_2S$  by reduction of  $La_2(SO_4)_3$  with  $H_2$  at 800 °C.

If the treatment with  $H_2S$  is carried out at a lower temperature (580-600°C), the sulfates of La, Ce and Pr form polysulfides  $Ln_2S_4$ , which decompose above 600°C to  $Ln_2S_3$  and S.

The anhydrous rare earth sulfates start to decompose above 600°C, yielding the basic sulfates  $Ln_2O_3 \cdot SO_3$ , whose decomposition temperatures decrease from La (1150°C) to Yb (900°C).

PROPERTIES:

Light yellow to light orange, opaque, hexagonal prisms. d<sup>25</sup> 4.86.

**REFERENCES:** 

- I. W. Klemm, K. Meisel and H. U. von Vogel. Z. anorg. allg. Chem. <u>190</u>, 123 (1930).
- II. O. Brill. Z. anorg. allg. Chem. <u>47</u>, 464 (1905); W. Biltz. Ibid. <u>71</u>, 424 (1911); Ber. dtsch. chem. Ges. <u>41</u>, 3341 (1908).

### Lanthanum Selenides

La<sub>2</sub>Se<sub>3</sub>, La<sub>2</sub>Se<sub>4</sub>

 $La_2O_3 + 3 H_2Se = La_2Se_3 + 3 H_2O$ 325.8 242.9 514.7 54.0

I. Both  $La_2Se_3$  and  $La_2Se_4$  are prepared by high-temperature reaction of the oxide or chloride with  $H_2Se_4$ .

A boat with  $La_2O_3$  is placed inside a quartz tube surrounded by an electric furnace. Several boats containing Se are placed ahead of the oxide and heated with Bunsen burners to a temperature at which the Se slowly evaporates. A stream of carefully purified  $H_2$  is passed through the quartz tube. After treatment for about five hours, during which the temperature of the  $La_2O_3$  is slowly raised from 600 to 1000°C,  $La_2Se_4$  is obtained in quantitative yield.

Heating the polyselenide for 30-60 minutes in high vacuum at 600-800 °C yields  $La_2Se_3$ . This operation must be carried out in a porcelain or corundum boat, since quartz reacts to form the rare earth oxide and SiSe<sub>2</sub>.

The same procedure is used for  $Ce_2Se_4$  and  $Pr_2Se_4$ . However, Nd yields only Nd<sub>2</sub>Se<sub>3.5</sub>. The other rare earths do not form polyselenides. The sesquiselenides of these elements are best prepared by treating the rare earth chlorides with H<sub>2</sub>Se:

$$2 La + 3 Se = La_2 Se_3$$
  
277.8 236.9 514.7

II. Synthesis from the elements by heating a stoichiometric mixture in a silicon carbide crucible held in an evacuated, sealed quartz tube.

#### PROPERTIES:

La<sub>2</sub>Se<sub>3</sub>: Brick red. Insoluble in both cold and boiling water; violently evolves  $H_2$ Se in dilute acids; decomposes slightly after several days in moist air. d<sup>20</sup> 6.19.

**REFERENCES:** 

I and II: W. Klemm and A. Koczy. Z. anorg. allg. Chem. <u>233</u>, 86 (1937); A. Koczy. Thesis, Danzig, 1936.

# La, Ce, Pr and Nd Monochalcogenides

#### LnS, LnSe, LnTe

These compounds are prepared by synthesis from the elements. The rare earth metal powder, as pure as possible, is placed in one of the arms of an L-shaped glass tube. The other arm contains the stoichiometric quantity of S, Se or Te (1:1 ratio). The tube is melt-sealed in vacuum and heated in an electric furnace until the non-metal is completely consumed. The temperature should reach 400-450°C by the end of 2-3 days. Powder pattern analysis of the products indicates the formation of nonhomogeneous materials containing  $Ln_2X_3$  and  $Ln_2X_4$ , but not LnX, which starts to form at 1000-1100°C.

For this reason, the samples obtained at the lower temperature are compressed (10 tons/cm.<sup>2</sup>) to cylindrical tablets in an atmosphere of  $CO_2$  and sealed (under vacuum) in quartz tubes. The material is then slowly heated to 1000°C in an electric furnace (to 1100°C in the case of the tellurides). The products are 99.2-99.5% pure.

In addition, CeS may be prepared by heating  $Ce_2S_3$  to 2200°C with a small excess of  $CeH_3$ ; an evacuated molybdenum container is used.

#### **PROPERTIES:**

The monosulfides of La, Ce, Pr and Nd are greenish yellow, the monoselenides reddish yellow, the monotellurides blue violet. The sulfides decompose in moist air to form  $H_2S$ . Crystal structure: type B1 (NaCl type).

REFERENCES:

A. Jandelli. Gazz. Chim. Ital. <u>85</u>, 881 (1955); E. O. Eastman, L. Brewer, L. A. Bromley, P. W. Gilles and L. N. Lofgren. J. Amer. Chem. Soc. <u>72</u>, 2249 (1950).

### Europium (II) Chalcogenides

EuS, EuSe, EuTe

 $EuCl_2 + S (Se; Te) + H_2 = EuS (EuSe; EuTe) + 2 HCl$ 222.8 32.1 (79.0; 127.6) 2,0 184.0 (230.9; 279.5) 72.9

A mixture of  $EuCl_2$  with a several fold excess of S, Se or Te is heated for several hours to 600 °C in a fast stream of purified H<sub>2</sub>.

This produces the desired chalcogenides. The S, Se or Te in excess of the desired composition is removed by heating for a few hours more at  $820^{\circ}$ C in the stream of H<sub>2</sub>.

Europium also forms an oxide, EuO, which may be prepared by heating  $Eu_2O_3$  with La or C.

PROPERTIES:

EuO: dark red or blue depending on the conditions of preparation; d 7.7. EuS: blue black; d 5.7. EuSe: brown black; d 6.4. EuTe: black, metallic appearance. All the Eu (II) chalcogenides crystallize in type B1.

REFERENCES:

W. Klemm and H. Senff. Z. anorg. allg. Chem. <u>241</u>, 259 (1939);
H. A. Eick, N. C. Baenziger and L. Eyring, J. Amer. Chem. Soc. <u>78</u>, 5147 (1956); M. Guittard and A. Benacerraf. Comptes Rendus Hebd. Séances Acad. Sci. <u>248</u>, 2589 (1959); L. Domange, J. Flahaut and M. Guittard. Ibid. <u>249</u>, 697 (1959); J. C. Achard. Ibid. <u>250</u>, 3025 (1960).

# Rare Earth Sulfates

 $Ln_2(SO_4)_3 \cdot nH_2O$ 

The oxide (0.3 g.) is dissolved in 20 ml. of hot 6N  $H_2SO_4$ . The solution is filtered and allowed to crystallize over conc.  $H_2SO_4$  in a vacuum desiccator. The product is filtered through fritted porcelain, washed twice with 10 ml. of water and once with 10 ml. of ethanol, and dried in air for four hours.

The product obtained from La by this procedure is  $La_2(SO_4)_3 \cdot 9H_2O$  and from Yb it is  $Yb_2(SO_4)_3 \cdot 11H_2O$ . The remaining rare earths and yttrium yield octahydrates.

Cerium sulfate,  $Ce_2(SO_4)_3 \cdot 5H_2O$ , is prepared by heating 3 g. of the chloride with 5 ml. of conc.  $H_2SO_4$  until all the hydrogen chloride has been removed. Then 20 ml. of water is added and the product is allowed to crystallize in a desiccator.

Alternate method: A neutral or slightly acid solution of the sulfate is treated with about 3/4 of its volume of ethanol. The sulfates may thus be isolated rapidly and quantitatively, without the evaporation stage.

Anhydrous rare earth sulfates may be prepared by dehydration of the hydrates at 400-600 °C. The same procedure can also be used with the acid sulfates obtained by evaporating the chlorides with conc.  $H_2SO_4$ .

1156

PROPERTIES:

The rare earth sulfates usually crystallize as octahydrates. The anhydrous sulfates are formed in the range of 155 to 295 °C; if the dehydration is carried out carefully, it is sometimes possible to detect intermediate stages, such as pentahydrates (Pr, Nd, Er) and dihydrates (La, Ce, Nd, Yb).

**REFERENCES:** 

W. W. Wendlandt. J. Inorg. Nuclear Chem. 7, 51 (1958); W. Biltz.
 Z. anorg. Chem. <u>17</u>, 427 (1911); O. Brill. Z. anal. Chem. <u>47</u>, 464 (1905).

#### **Rare Earth Nitrides**

#### LnN

Rare earth nitrides may be prepared by heating the metal in a stream of  $N_2$  or  $NH_3$ , or by reaction of the chloride with  $NH_3$ . The preparation of LaN by the first method is given as an example.

LANTHANUM NITRIDE, LaN

Lanthanum filings (several hundred milligrams), prepared from the metal in a stream of  $N_2$ , are freed of iron with a magnet and heated in a molybdenum boat placed in a stream of purified  $N_2$ . The azotization requires 2-4 hours at 750°C, 1-2 hours at 900°C.

The nitrides of Ce, Pr, Nd, Sm, Eu and Yb may be prepared by a basically similar method.

PROPERTIES:

Black powder; evolves  $NH_3$  in moist air. Crystal structure: type B1 (NaCl type).

**REFERENCES:** 

B. Neumann, C. Kröger and H. Haebler. Z. anorg. allg. Chem. 207, 148 (1932); W. Muthmann and H. Kraft. Liebigs Ann. 325, 274 (1902); A. Jandelli and E. Botti. Atti R. Acad. Naz. Lincei, Rend. [6] 25, 129 (1937); R. A. Young and W. T. Ziegler. J. Amer. Chem. Soc. 74, 5251 (1952); H. A. Eick, N. C. Baenziger and L. Eyring. Ibid. 78, 5987 (1956); B. M. Ormont and E. V. Balabanovich. Russian Patent 51,424, Chem. Zentr. 1938, II, 573; W. Klemm and G. Winkelmann. Z. anorg. allg. Chem. 288, 87 (1956).

### **Rare Earth Nitrates**

### Ln(NO<sub>3</sub>)<sub>3</sub> (anhydrous)

 $Ln_2O_3 + 6N_2O_4 = 2Ln(NO_3)_3 + 3N_2O_3$ 

Anhydrous nitrates of the rare earths may be obtained from the oxides by heating with  $NH_4NO_3$  or, better, by treatment with liquid  $N_2O_4$ . However,  $Nd(NO_3)_3$  can be prepared only from  $Nd_2O_3$ and  $N_2O_4$ ; heating  $Nd_2O_3$  with  $NH_4NO_3$  yields  $Nd(NO_3)_3 \cdot NH_4NO_3$ .

The apparatus for the preparation from the oxides and  $N_2O_4$  is shown in Fig. 289. Drying tower *a*, filled with  $P_2O_5$ , is connected to storage bottle *e* through a vacuum-type stopcock *b*. A mercury manometer, which serves as a safety valve, is attached at *f*, and a McLeod gage is connected to *g* via a cold trap. The 150-ml. steel bomb *h* is equipped with a needle valve at the top and a square thread screw at the bottom; the latter is for the introduction of the dry oxide (about 2 g.) and removal of the reaction product. The bottom neck of the bomb and the corresponding surface of the screw head have machined seats for a lead O-ring. Lead packing may also be used at the junction of the bomb and the needle valve. The metal and glass tubes are connected at *i* by means of a cement seal (for example, Glyptal). Two stopcocks, *c* and *d*, and a cold trap *k* complete the system.



Fig. 289. Preparation of anhydrous rare earth nitrates. a drying tower; b, c, d vacuum-type stopcocks; f, g connections to manometers; h steel bomb; k cold trap.

The apparatus is evacuated through l to about 0.02 mm. Stopcock c is closed and about 30 ml. of liquid N<sub>2</sub>O<sub>4</sub> is condensed in eby cooling with liquid N<sub>2</sub>. Then c is opened, b and d are closed, and the N<sub>2</sub>O<sub>4</sub> is distilled into the steel bomb h (40°C water bath at e, cooling with liquid N<sub>2</sub> at h). The needle valve is closed and the bomb disconnected at *i*. A steel jacket is screwed on and the bomb is heated for 24 hours at 150°C. After cooling, the  $N_2O_4$  is removed (vacuum) via a system of drying towers filled with  $Mg(ClO_4)_2$  and collected in a trap cooled with Dry Ice-acetone.

The last traces of  $N_2O_4$  are removed by heating the product in a drying pistol at 137°C (boiling xylene). Very pure nitrates are obtained in 100% yield.

Up to now, this method has been used for the preparation of the nitrates of Y, La, Pr, Nd, Sm and Gd. For the preparation of  $(NH_4)_2Ce(NO_3)_5$ , see G. F. Smith, V. R. Sullivan and G. Frank, Ind. Eng. Chem., Anal. Edit. 8, 449 (1936), as well as p. 1133 f. of this Handbook.

PROPERTIES:

Loose powders; form clear solutions with water and ethanol (very exothermic process). The nitrate colors differ only slightly from those of the corresponding anhydrous chlorides.

REFERENCE:

T. Moeller and V. D. Aftandilian. J. Amer. Chem. Soc. <u>76</u>, 5249 (1954);
T. Moeller, V. D. Aftandilian and G. W. Cullen in: W. C. Fernelius, Inorg. Syntheses, Vol. V, New York-London, 1957, pp. 37-42;
L. F. Audrieth, E. E. Jukkola, R. E. Meints and B. S. Hopkins. J. Amer. Chem. Soc. <u>53</u>, 1807 (1931).

### Rare Earth Cyclopentadienides

### Ln(C<sub>5</sub>H<sub>5</sub>)<sub>3</sub>

 $LnCl_3 + 3 NaC_5H_5 = Ln(C_5H_5)_3 + 3 NaCl$ 

The anhydrous rare earth chloride, in tetrahydrofuran solution, is treated (stirring) with the stoichiometric quantity of cyclopentadienylsodium. The solvent is then removed by distillation and the product is sublimed at 200-250 °C in vacuum ( $10^{-4}$  mm.).

Up to now, only the tricyclopentadienides of Sc, Y, La, Ce, Pr, Nd, Sm and Gd have been prepared.

PROPERTIES:

Crystalline compounds; begin to decompose above 400°C. Insoluble in hydrocarbons, soluble in tetrahydrofuran and 1,2-dimethoxyethane. Decompose in water to cyclopentadiene and the hydroxide. Quite stable in air, with the exception of  $Cd(C_5H_5)_3$ .

**REFERENCE:** 

G. Wilkinson and J. M. Birmingham. J. Amer. Chem. Soc. <u>76</u>, 6210 (1954).

# SECTION 22

Titanium, Zirconium, Hafnium, Thorium

### P. EHRLICH

### Titanium

#### Ti

Due to its great affinity for a large number of elements, the preparation of titanium poses considerable difficulties. In particular, N, C and O dissolve to an appreciable extent in the metallic phase, and cause cold-shortness even when present in minute quantities. They cannot be removed either chemically or by sintering or melting in high vacuum. Consequently, the relatively easy conversion of  $TiO_2$  with Ca (method I below) yields only 98% pure metal, even under conditions where the highest purity of apparatus and raw materials is maintained.

Pure metal that is ductile while cold can therefore be prepared only by methods which use halides as the starting materials. However, these procedures, which are based on the reactions  $K_2 \text{TiF}_6$  (or  $\text{Na}_2 \text{TiF}_6$ ) + Na (method II) or  $\text{TiCl}_4$  + Na (method III below), suffer from the drawback that the deposited metal is usually porous or flaky, which leads to reoxidation during the removal of the alkali halide by-product; it is therefore used only as a crude starting material for the purification process. Nevertheless, careful adherence to a number of precautionary measures permits the preparation, even by these methods, of pure metal which can be cold-worked. The Kroll magnesium process (method IV), which utilizes the reaction between TiCl<sub>4</sub> and Mg, is used at present both in the laboratory and in industry.

The highest purity (0.03% C and  $\sim 0.006\%$  N) is attained via the elegant recovery process of van Arkel and de Boer (method V below). This is based on the thermal decomposition of titanium iodide at 1100-1500°C.

I. PREPARATION OF CRUDE METAL FROM THE OXIDE AND CALCIUM

$$TiO_2 + 2Ca = Ti + 2CaO$$
  
79.9 80.2 47.9 112.2

When only crude starting metal for the refining process is desired, the preparation may be simplified and carried out in a bomb made of two steel sections welded together. Section 1 (the one of larger diameter) consists of a tube of type 304L stainless or low-carbon steel with a welded-on bottom. A wall thickness of 1 mm. is sufficient if the tube diameter does not exceed 25 mm. This section is annealed at 1000°C in moist H<sub>2</sub> (for more efficient removal of the P and C present). A second, crucible-like section, 40-60 mm. long, of exactly the same shape and precisely fitting into the first section (in such a way that the two sections telescoped together make up a vessel closed on all sides), is charged with the starting materials and forced as far as possible into the first section in order to reduce the air space inside to the minimum and to give the tightest possible seal between the two walls. If this is done, then the rims of the two tubes may be welded together without a welding rod; the lower section of the tube, that is, the section encompassing the charge, is cooled in water during the welding operation. One can avoid, to a large extent, the penetration of the welding gases into the bomb either by extending the sealing surface between the two tubes (that is, by using longer tubes for an identical charge), or by crimping the upper rim of the outer crucible around the inner one. As an explosion protection, and to provide a backup to strengthen the bomb walls, a closely fitting external tube or jacket, made of the same material and open at both ends, should surround the bomb. Scaling of the bomb may be prevented by preheating the latter inside a porcelain tube in a stream of H<sub>2</sub>; if the heating must be carried out in air, a coat of aluminumbronze paint will prevent too rapid oxidation of the tube.

Following the reaction (see description below) the bomb is allowed to cool completely before being sawed open. The sawing should not introduce any iron filings into the product (avoid tilting the bomb during sawing or cutting at an angle). The reaction product can usually be loosened by gentle tapping with a hammer while the tube is clamped in a vise. Alternatively, the crucible may be sawed open along its length and the steel jacket is just peeled off.

Only Si-free, well-dried TiO<sub>2</sub> starting material should be used, to avoid formation of silicon or silicides. If this precaution is not taken, these impurities are carried over in the subsequent refining process and are incorporated into the titanium ingot. Furthermore, the reduction should be carried out only with distilled Ca; addition of distilled Na is advantageous. Thus, the reduction of 25 g. of TiO<sub>2</sub> with a mixture of 40 g. of Ca and 20 g. of Na yields about 13 g. of crude Ti (with a metal content of about 90%). Heating for 20 minutes at 1000 °C suffices for complete reduction. After cooling and opening the tube, the contents are ground to pea size and leached with alcohol, water and increasingly concentrated portions of hydrochloric acid. The residue is washed free of chloride, the water is removed with alcohol, and the product is dried at 110°C.

The preparation of titanium that is malleable when hot (>200°C) by this process is described by Kroll. Pure TiO<sub>2</sub> (770 g.), turnings of distilled Ca (770 g.), and fused and pulverized CaCl\_/BaCl\_ (750 g./250 g.) are mixed and pressed into briquets, which are allowed to react under 99.2% Ar in an electric furnace at > 700°C. The addition of the salts is necessary to moderate the reaction and, above all, to prevent the formation of CaTiO<sub>3</sub>, a product which does not react with Ca even on repeated reduction. The use of CaH<sub>2</sub> in the second reduction has proved useful, since the powdery hydride mixes very readily with the other reactants while the nascent H<sub>2</sub> it evolves is a powerful reducing agent. Thus, 348 g, of Ti (from the first reduction stage) + 400 g, of CaCl<sub>2</sub>/BaCl<sub>2</sub> (3:1) + 50 g. of Ca + 50 g. of CaH<sub>2</sub> gave a yield of 337 g. of metal after heating for one hour at 1000°C under 99.6% Ar. The very well-sintered product is crushed and washed with water and concentrated hydrochloric acid, yielding fairly homogeneous granules.

The original reference covers the constructional details of the furnace.

With sufficiently small inputs  $(20-30 \text{ g. of TiO}_2)$  the second reduction may also be carried out in the welded bomb and without the addition of CaH<sub>2</sub>; the temperature should then be 1000°C (see also the procedure for Th, method II).

As in the preparation of the rare earth metals [F. H. Spedding et al., Ind. Eng. Chem. <u>44</u>, 553 (1952)] the addition of free iodine to the reduction mixture is recommended, since the large heat of formation of  $CaI_2$  facilitates fusion of the metal.

### II. PREPARATION OF CRUDE METAL FROM FLUORIDES AND SODIUM

or  $Na_{2}TiF_{6} + 4Na = Ti + 6NaF$   $207.9 \quad 92.0 \quad 47.9 \quad 252.0$   $K_{2}TiF_{6} + 4Na = Ti + 2KF + 4NaF$   $240.1 \quad 92.0 \quad 47.9 \quad 116.2 \quad 168.0$ 

The fluorotitanates are prepared by dissolving pure  $TiO_2$  in an excess of warm 20-40% hydrofluoric acid, treating the mixture with a stoichiometric quantity of NaOH or KOH, evaporating the solution without overheating (at 40-60°C) until saturation, and allowing the product to crystallize. In the case of the potassium salt, the product is the monohydrate; it is readily recrystallized from water. Heating of the air-dried product for two days at 35°C readily yields the anhydride, which in air at 500°C decomposes to the oxyfluoride. The Na<sub>2</sub>TiF<sub>6</sub> crystallizes already as the anhydrous salt and may be obtained in 99.9% purity by repeated precipitation with alcohol from aqueous solution. The small amount of water remaining in the product after drying in air is difficult to remove by heating without causing partial hydrolytic decomposition [H. Ginsberg and G. Holder, Z. anorg. allg. Chem. <u>190</u>, 407 (1930); <u>196</u>, 188 (1931); <u>201</u>, 198 (1931); <u>204</u>, 225 (1932)].

The Na<sub>2</sub>TiF<sub>6</sub>, in portions of up to 1 kg., may be readily reduced with a 10% excess of Na in a bomb. The sodium is cut into small cubes and mixed with the hexafluorotitanate. After filling, the bomb is welded as in method I and heated to 1000°C. It is imperative that the fluoride be absolutely dry, otherwise an explosion may occur.

When  $K_2 TiF_6$  is used in the same process, a Na excess of only 1% is used, to prevent the formation of too much K, which may cause ignition of the mixture upon opening of the tube.

The one great disadvantage of this process is the fact that removal of the fluorine from the product requires very long boiling with large quantities of water. Direct washing of the fluorine-containing reaction mass with hydrochloric acid is not feasible, since the alkali fluorides react with the acid to form hydrofluoric acid, which dissolves the titanium metal. On the other hand, boiling with water results in considerable oxidation: the Ti thus produced may contain more than 20% of the oxide. After the treatment with water, the metal is boiled a few times with aqueous sodium hydroxide and is then treated with cold, dilute hydrochloric acid (too much Ti goes into solution with warm acid).

III. PREPARATION OF CRUDE METAL FROM CHLORIDE AND SODIUM

 $TiCl_4 + 4Na = Ti + 4NaCl$ 189.7 92.0 47.9 233.8

If the reagent quantities are small, the welded steel bomb described in method I can be used. The temperatures must be very high (to start the reaction, the bomb must be red-hot) and thus the  $TiCl_4$  vapor pressure is very high. Larger quantities (500 g. of  $TiCl_4 + 245$  g. of Na) must therefore be heated in a thick-wall steel bomb, the lid of which is sealed on with a copper gasket and secured with a heavy screwed-on cap.

The TiCl<sub>4</sub> pressure in the bomb can be kept low by one of two methods: *a*) the starting temperature of the reaction may be lowered by the addition of a tablet of  $KClO_3$ -Na; the small amount of oxygen introduced is not detrimental provided only crude metal is desired; *b*) the reaction tube may be constructed in such a way that there exists a temperature gradient and only the sodium is heated to 700-800 °C. If the amount of starting material is not too small, the heat of reaction developed in the process is sufficient to cause sintering of the metal; the heat may even be sufficient for partial melting of the charge. The product titanium is first washed with alcohol to remove the excess sodium, then with water to remove salts, and finally with dilute hydrochloric acid. After repeated washing with water, alcohol and ether, the metal is dried in a vacuum desiccator. Assuming the above-mentioned charge of 500 g. of TiCl<sub>4</sub>, the product consists of about 31.5 g. of half-fused metal and 4.5 g. of fine powder, as well as 71 g. of lumps and grains whose Ti content ranges between 96 and 99.5%. The powder fraction oxidizes quite readily.

This metal is much better suited as crude Ti for the refining process than the product obtained from the hexafluorotitanate.

In the industrial Degussa process, 46 kg. of Na is heated to 700-800°C. Then, 85 kg. of TiCl<sub>4</sub> is piped onto a layer of molten KCl/NaCl (15/15 kg.) situated below the Na. The resultant metal consists of 98% Ti and 2% Fe.

### IV. KROLL MAGNESIUM PROCESS

### A) PREPARATION OF THE METAL

Magnesium works just as well in the reduction of  $TiCl_4$  as sodium; in addition, commercial magnesium is already very pure and may be handled in air without special precautions. Thus, magnesium is the preferred reducing agent.

The reduction apparatus is shown in Fig. 290.

Since titanium attacks iron at high temperatures, the entire reaction zone of the crucible must be lined with a 1.5-mm.-thick sheet of molybdenum. Although molten Ti also adheres to molybdenum, the two metals can later be separated on a lathe. The TiCl<sub>4</sub> itself does not react at high temperatures with either Fe or Mo; the only precaution necessary is to keep all iron parts inside the furnace oxide-free.

The reaction crucible b, lined with molybdenum sheet c, is charged with 360 g. of very pure Mg blocks (the Mg metal surfaces are precleaned with a file). The adapter cover e, which carries the dropping funnel m and the Ca electrodes, is put in place and the entire system is evacuated to 0.1 mm. Very pure Ar is introduced, and an electric arc is struck and maintained for 10 min. between the two Ca electrodes o; the resultant Ca vapor serves as a scavenger for moisture and impurity gases. Final drying of the Ar is achieved by dropwise addition of a small quantity of TiCl<sub>4</sub> from the small dropping funnel m.



Fig. 290. Preparation of titanium metal from titanium (IV) chloride and magnesium. a) chrome-nickel alloy outer container (Inconel, Nichrome, or other similar alloys may be used); b) iron crucible; c) molybdenum lining; d) ironplate lid; e) adapter cover; f) cooling chambers; g) cooling coils (lead); h) vacuum line; i) rubber balloon for Ar; k) rubber connections and seals; l) iron inlet tube for  $TiCl_4$ ; m) dropping funnel with  $TiCl_4$ ; n) electrodes; o) calcium rods; p) glass adapter for a sight glass; q) stopcocks; r) storage bottle with TiCl<sub>4</sub>; s) CaCl<sub>2</sub> tube.

When the alloy container g reaches a temperature of about 700°C, 500 ml. of TiCl<sub>4</sub> is slowly added to the reaction chamber from the storage bottle r. The addition rate is such that it takes 1.5 hours to add all of the TiCl<sub>4</sub>. The temperature, which rises to about 1050°C, should be precisely controlled during the entire addition. The remaining 150 ml. of TiCl<sub>4</sub> is then added very slowly over a period of 0.5 hour, the temperature being gradually raised above the boiling point of Mg (to a maximum of 1180°C).

The molten Mg creeps over the surface of the nascent clumps of Ti, thus constantly contacting fresh  $TiCl_4$ . In the process small quantities of Mg and  $MgCl_2$  are occluded in the Ti; the final heating of the iron crucible to above the boiling point of Mg is intended to counteract this phenomenon.

The progress of the reaction may be observed through the quartz window set in adapter p. If the rate of addition of TiCl<sub>4</sub> is not precisely controlled, the inlet tube l may become plugged with Ti sponge.

After cooling in argon, the crucible is full of large clumps of light Ti metal embedded in white MgCl<sub>2</sub> crystals. The metal contains extremely finely divided Mg and MgCl<sub>2</sub>; however, no Mg-Ti alloy is formed. This mass is removed from the crucible with the help of a lathe, cutting as far as the molybdenum lining; the pieces of Ti are held so firmly in the surrounding MgCl<sub>2</sub> that metal turnings can be produced without any difficulty. These are first very carefully leached with water, and are then treated with an excess of dilute HCl. Decantation yields about 1% of the product in colloidal form. The smaller turnings are wet-ground in a ball mill and worked up separately. They are unsuitable for the production of ductile Ti. The coarser pieces are crushed to 10-12 U. S. mesh size, and this coarse metal powder is rewashed. separated from the fines, and etched with hot hydrochloric acid (1:3) until the acid becomes deep violet. The acid treatment is necessary because the crushing operation oxidizes the surface of the metal particles (in contrast to the zirconium oxides, the titanium oxides can be removed by acid leaching). After another washing procedure, first in cold 5% hydrochloric acid and then in water, followed by drying, the powder is freed of Fe with a magnet, rescreened, washed with alcohol and dried at 120°C. The yield of Ti metal is 284 g. (95.9%).

Worner, as well as Wartmen et al., has modified the above procedure in several respects. They carry out the reaction in vacuum; however, this necessitates the use of a double-walled container. The addition of TiCl<sub>4</sub> is carried out much more rapidly (80% at 30-40 ml./min., the remainder at 10 ml./min.), so that 1500°C temperatures occur locally, and external heating may be stopped as early as five minutes after the start of the TiCl<sub>4</sub> addition. When the reaction is complete, heating at 900°C is continued for 45 minutes. The reaction product is not leached; MgCl<sub>2</sub> and unreacted Mg are partly removed by evaporation and partly by fusing and draining.

# B) REMELTING OR RESINTERING OF THE METAL

The Ti sponge may be converted to solid metal by fusion in an arc furnace, in which either high vacuum or a pure argon (99.92%) atmosphere is employed; the other acceptable procedure is sintering with alternating pressing and heating in high vacuum ( $10^{-4}$  mm.) at 1000°C. The Ti powder may also be hot-rolled in air while contained in a welded steel container. In the last case, contamination with Fe is slight and the iron is easily removed by etching the ingot after unwinding the steel sheet. For further details, see the original references.

Assuming that the proper conditions are observed, the product metal is about 99.8% pure, and contains 0.06% Fe, 0.1% O, 0.02% N and 0.02% MgCl<sub>2</sub>.

### V. THE REFINING PROCESS OF VAN ARKEL AND DE BOER

$$TiI_4 = Ti + 2I_2$$
  
555.6 47.9 507.7

a) The iodides are used for the preparation of small quantities ( $\sim 20-30$  g.) of metal; these highly hygroscopic compounds are



Fig. 291. Preparation of titanium metal by the process of van Arkel and de Boer. g) pyrex bulb; a, b, c) triangular arrangement of tungsten bus bars;  $d_1, d_2$ ) tungsten wires; h) iodine storage flask; l) shatter valve; n) steel ball.

not introduced directly as raw materials, but are produced as intermediates during the process in which they form from crude metal and iodine. The most suitable crude titanium for this process is that prepared from TiCl<sub>4</sub> and Na. Titanium oxide, nitride or carbide are attacked by the iodine; thus, the corresponding nonmetals are left unchanged and do not incorporate into the growing metal ingot. The weak point of this refining process is that a considerable number of other metals (e.g., Zr, Hf, Th, V, B, Si, as well as Al and Fe if the filament temperature is low) are codeposited with the desired titanium; therefore, these impurities should be removed during the preparation of the crude metal, that is. prior to refining.

The Pyrex thermal decomposition flask is shown in Fig. 291. The tungsten bus bars a, b and c(diameter of each 6 mm.) are arranged in a triangular pattern and sealed into the glass. The Ti de-

posits on drawn tungsten core wires  $(d_1 \text{ and } d_2)$ , each 0.04 mm. thick and 400 mm. long.

These wires cannot just simply be stretched directly between the electrodes, as would appear from the drawing. If this were done, the rapid rate of heat conduction through the tungsten rods could cause excessive cooling of the wire ends and consequently prevent the titanium from depositing at the cold spots. This would result in nonuniform deposition, that is, preferential accumulation of the metal on the glowing sections, which would thus become heavier and unbalance the entire wire. Consequently, after a certain time has elapsed, the slightest mechanical shock would be sufficient to break the thin wires and interrupt electrical contact. The critical spots e and f should therefore be strengthened by insertion of reducer pieces made of progressively thinner tungsten wires. The simplest arrangement consists of a 1-mm.-thick wire ring fixed in a slit in the bus bar; this ring, in turn, is fitted with a drilled 0.2-mm. ring to which the glowing wire is attached.

Crude Ti (40 g.) is placed in g and 12 g. of I<sub>2</sub> in evacuated space h. The flask is evacuated to  $< 10^{-3}$  mm. and the metal is degassed by heating to about 500°C. At the same time, the entire glass part of the apparatus (except for the iodine tube) is dried and degassed by fanning with a flame while the reduced pressure is maintained. As soon as a sufficiently high vacuum has been restored, a predetermined starting current (about 0.25 amp.) is applied to the tungsten wires to bring them to a "black body temperature" of 1400°C, as measured by an optical pyrometer. The system is cooled and sealed at i, the thin glass partition l is broken by means of an electromagnet and steel ball n, and the  $I_2$  from h is allowed to flow into g. A temperature of 200°C is sufficient for a rapid reaction of the iodine with a portion of the Ti, a reaction sometimes even accompanied by the appearance of a flame. Following this the apparatus is melt-sealed at k and heated to  $550^{\circ}$ C in a furnace; the tungsten wire  $d_1$  is then brought to a glow at the same current as above. The apparent temperature read on the pyrometer is now lower because of the colored vapors rising from the material. This temperature must be held constant during the entire subsequent procedure (by increasing the current as the thickness of the deposited Ti layer increases).

The equilibria prevailing in the system are highly temperaturedependent. Furnace temperatures below  $250^{\circ}$ C produce TiI<sub>4</sub>, which then decomposes on the hot wire in accordance with the above equation. At higher temperatures, TiI<sub>4</sub> reacts with the crude Ti to form TiI<sub>2</sub>, which has a considerably lower vapor pressure. Only at temperatures above  $500^{\circ}$ C does this pressure become large enough to again produce titanium deposition on the glowing wire. The Ti metal formed at higher temperatures is so free of iron that the latter cannot even be detected.

Gases which may still be present in the flask during the refining of the Ti are bound by the metal (thus, a small amount of nitride is often formed). For this reason, the current to the first wire is shut off after a certain time and that to the second wire,  $d_2$ , is turned on; the Ti metal which then deposits on  $d_2$ is completely pure. Because of the gradual build-up of titanium, the current must be raised up to 200 amp. when the ingot reaches a thickness of about 5 mm. This requires about 24 hours; if the starting material is the crude Ti obtained from the hexafluorotitanate, the build-up rate is lower.

The current to the furnace can be gradually reduced to zero in the course of the run, since the growing Ti rod begins to radiate enough heat to maintain the required temperature throughout the system (toward the end of the run, it may even be necessary to cool the furnace space with air).

If the crude metal used in the refining process is prepared from pure starting materials, the resultant smooth Ti rod is almost completely pure, since the tungsten wire substratum constitutes only about 0.01% of the rod. The metal has about the same ductility as Cu, and may therefore be cold-worked and rolled.

The following method is well suited for the conversion of a piece of ductile titanium (it applies also to Zr or Hf) to powder: Ti is treated at 600°C in a stream of H<sub>2</sub>. The gas must be extremely pure (see p. 111 ff.). The resultant hydride is brittle and easily ground to a powder. The H<sub>2</sub> may then be removed by heating in high vacuum at 1000°C.

b) This method has recently been used in the U.S. to prepare 700-g. quantities of Ti. The operation is carried out in Pyrex containers 900 mm. long and 200 mm. in diameter, but metal tubes have also been used with great success. For a given size of tube, the metal tubes are much easier to handle and simpler to cool. They contribute to the safety of the operation since an oil bath can then be used. The crude Ti is not placed at the bottom of the tubular vessel but is held in a layer 10-15 mm. thick at the walls by means of a cylinder of perforated Mo sheet. With this arrangement, it is also possible to dispense with the additional furnace heating. The glowing wires, in the form of hairpins, are hung from three tungsten rods; if one wire burns out, two more are still available. Titanium prepared in this manner contains 0.03% C, 0.003% N, 0.002% O, 0.04% Si, 0.04% Fe, 0.05% Al and 0.002% S.

### VI. PREPARATION BY ELECTROLYSIS OF MELTS

Because of its high melting point, the Ti formed by electrolysis deposits on the cathode in the form of a solid cluster impregnated with the melt. The presence of even minute traces of moisture or oxygen causes the deposition of a finely crystalline material with a high salt content; suitable operating techniques, however, make it possible to obtain large crystals of pure metal.

a) Crude titanium may be obtained by electrolysis of a solution of TiO or mixed TiO-TiC crystals in a CaCl<sub>a</sub> melt at 700-850°C.

b) The electrolytic decomposition of  $K_2 TiF_6$  in a bath of NaCl, on the other hand, yields a very pure, coarsely crystalline metal. In this process the melt becomes enriched in NaF, according to the overall equation

$$K_{2}TiF_{6} + 4 NaCl = 2 KF + 4 NaF + Ti + 2 Cl_{2}$$

while chlorine is evolved on the anode.

c) Another process uses an electrolysis cell in which the cathode chamber is separated from the anode by a diaphragm of sintered alumina; TiCl<sub>4</sub> vapor is introduced into a melt of alkali metal chloride or alkaline earth chloride in the cathode chamber. The resultant dissolved lower chlorotitanates decompose to the metal at a later stage of the electrolysis.

d) Metal of very high purity may be recovered from crude titanium or titanium waste by anodic solution of the starting material in a melt of alkali metal chloride containing a small amount of lower titanium chlorides, and reprecipitation of the Ti at the cathode.

VII. REDUCTION OF TiO<sub>2</sub> WITH CaH<sub>2</sub>

A mixture of TiO<sub>2</sub> and CaH<sub>2</sub> (in 40% excess) is heated for one hour in hydrogen at atmospheric pressure (electric furnace, 950-1075 °C); the product is treated with dilute hydrochloric acid. A fine powder, with a metal content of 96%, is obtained; the remainder is mainly H<sub>2</sub> (3%). This process is also suitable for preparation of V, Nb and Ta from their oxides.

PROPERTIES:

Silvery white, ductile metal. M.p. 1730°C; d 4.45. Crystal type A3. Hexagonal  $\alpha$ -Ti converts at 885°C to the body-centered cubic form ( $\beta$ -Ti). Electrical resistivity  $\rho = 42 \cdot 10^{-8} \Omega \cdot \text{cm}$ .

Scarcely or not at all attacked by acids and bases; dissolves readily in hydrofluoric acid.

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### Zirconium, Hafnium

### Zr, Hf

The general remarks concerning titanium apply equally well to zirconium and hafnium. These elements are thus prepared via the same methods and generally in the same equipment as described in detail in the section on titanium. In the following, we shall discuss only those details that differ from the above. Where Hf is not discussed separately, the conditions specified for Zr apply.

I. 
$$ZrO_2 + 2Ca = Zr + 2CaO$$
  
123.2 80.2 91.2 112.2

Crude starting Zr for the refining process is prepared via the sealed bomb method, with corresponding changes in the amounts of materials used, chiefly the addition of Na, which is highly recommended for the reduction of  $ZrO_2$  (e.g., 20 g. of  $ZrO_2 + 20$  g. of Ca + 10 g. of Na); heating to 1300°C produces a metal still containing oxygen. Although this causes cold-shortness, the metal becomes malleable somewhat above 200°C.

**II.** 
$$K_2 ZrF_6 + 4Na = Zr + 2KF + 4NaF$$
  
283.4 92.0 91.2 116.2 168.0

Crystalline  $K_2 Zr F_6$  (60 g.), prepared by cooling a heated solution of the hydroxide in KHF<sub>2</sub>, is heated with 27.5 g. of Na in a sealed bomb at 1200-1300°C. The resultant 18 g. of crude metal is a suitable material for the refining process.

III. 
$$ZrCl_4 + 4Na = Zr + 4NaCl_{233.0} = 92.0 = 91.2 = 233.8$$

Up to 1 kg. of sublimed  $ZrCl_4$ , crushed to lumps, and 450 g. of distilled Na may be reacted in one run in a sealed steel tube at 850°C. The bottom of the tube is charged with a layer of Na; this is followed by the reaction mixture  $(ZrCl_4 + Na)$ , topped with a layer of Na. Since the vapor pressure of  $ZrCl_4$  is considerably lower than that of TiCl<sub>4</sub>, the processing of larger quantities is simpler. The heat evolved in the reaction is so large that partial sintering of the metal occurs.

IV. 
$$ZrCl_4 + 2Mg = Zr + 2MgCl_2$$
  
233.0 48.6 91.2 190.4

As in the magnesium process for Ti, Zr may be prepared by reduction of  $ZrCl_4$  vapor with Mg in a helium atmosphere (see references below for further details).

V. 
$$ZrI_4 = Zr + 2I_2$$
  
598.9 91.2 507.7

The glowing wire temperatures during refining should be  $\sim 1400$  °C in the case of Zr and  $\sim 1600$  °C in the case of Hf. The Pyrex vessel must be kept at 600 °C. The crude metal obtained from the chloride is the most suitable raw material. Recently Zr has been prepared in large glass vessels in the form of rods weighing up to 200 g.

PROPERTIES:

Silvery white, ductile metals.

	Formula weight	M.p.	d	$\alpha$ (close-packed hexag.) $\rightarrow \beta$ (bcc.)	Resistivity
Zr	91 <b>.22</b>	1860°C	6.50	870°C	$41 \cdot 10^{-6} \Omega \cdot cm$ .
Hf	178 <b>.</b> 6	<b>22</b> 30°C	13.3	~1500°C	30 \cdot 10^{-6} \Omega \cdot cm.

Scarcely or not at all attacked by acids and bases; dissolve readily in hydrofluoric acid. For the pulverization of solid metals via hydrogenation and dehydrogenation (hydrides), see pp. 1170 and 1184.

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#### Thorium

#### Тh

Despite the close resemblance in chemical behavior of the metals and their compounds, thorium differs from titanium, zirconium and hafnium in one respect, and that makes the preparation of the metal much easier. Thus, although the affinity of thorium metal for O, N and C is large, its cubic face-centered lattice cannot accommodate these nonmetals in solid solutions. The result is that, even on incomplete purification, these nonmetals are present only in the form of compounds and in small quantities; they have thus little effect on, for example, the mechanical properties of the metal. For this reason Th may be obtained in a cold-workable form by pressing and sintering the powder.

In contrast to titanium and zirconium, the preparation of thorium metal via reduction of the oxide with calcium (method II) acquires increased importance and rivals the reduction of the tetrachloride with sodium (method I). Melt electrolysis (method III) is another possibility. Neglecting the small oxide content (up to 1%), which in any case has never been determined precisely, the metal obtained by any of the three methods is already quite pure and contains only 0.1-0.2% of other impurities. The Th prepared by the refining process (method IV), is definitely oxygen-free and should in any case yield the purest product.

### I. REDUCTION OF THE TETRACHLORIDE WITH SODIUM

$$ThCl_4 + 4 Na = Th + 4 NaCl_{373.9} 92.0 232.1 233.8$$

Up to 300 g. of oxychloride-free sublimed ThCl<sub>4</sub> may be reacted at 500°C with vacuum-distilled Na (25% excess) in a welded iron bomb (see method I, section on titanium). The iron crucible should be filled with the reaction mixture in the same way as in the reaction between  $ZrCl_4$  and Na, that is, layer by layer. Following the reaction, the bomb is completely cooled, opened and the reaction product treated, first with alcohol (to remove the excess Na), then with water (always maintaining the solution on the alkaline side). After complete removal of the chlorine, the residue is treated with 2N HNO<sub>3</sub> to dissolve any Th(OH)<sub>4</sub> which may be present, filtered with suction, thoroughly washed with water, alcohol and ether, and dried in vacuum at 300°C. The metal yield is 55%, in the form of lead-gray platelets and pellets. The coarsest particles are also the purest and contain 0.1% O (1% ThO<sub>2</sub>).

# II. REDUCTION OF THE OXIDE WITH CALCIUM

 $ThO_2 + 2 Ca = Th + 2 CaO$ 264.1 80.2 232.1 112.2

#### A) PREPARATIVE PROCESS

The process is based on the reduction of very pure ThO<sub>2</sub> with distilled Ca in the presence of anhydrous CaCl<sub>2</sub> heated to  $450^{\circ}$ C.

The CaCl<sub>2</sub> melts at the temperature of the reaction, affording a liquid reaction medium. The heavy Th product settles to the bottom and is thus protected by a layer of melt. The apparatus is either a steel bomb capped with a threaded conical lid (cf. the paper of Marden and Rentschler) or the simpler welded steel tube described under method I for the preparation of Ti. The charge, which is made up of four parts of  $ThO_2$ , four parts of CaCl<sub>2</sub> and three parts of ground Ca, is vigorously shaken in a closed bottle to achieve the most complete mixing possible. The bomb is filled rapidly, sealed and heated for one hour at 950°C. The tube is then cooled and opened; the resultant pea-sized reaction product is gradually added to water (about two liters per 40 g. of starting  $ThO_2$ ) with vigorous stirring to prevent a local temperature rise. After the calcium has completely reacted with the water and the evolution of gas ceases, stirring is stopped, the supernatant liquid is decanted, and the solid is washed four times with two-liter portions of water, vigorously shaking each time for 5-10 minutes. The decanted supernatants are lowconcentration suspensions of dark, fine Th. Finally, 200 ml. of water is added to the remaining heavy residue, followed by 25 ml. of conc. nitric acid (vigorous stirring). The odor of acetylene is noticeable, and if the ThO<sub>2</sub> used in the preparation was made from thorium nitrate which contained some sulfate, the odor of H<sub>2</sub>S will also be present. After 10 minutes, the solution is diluted tenfold, the product is allowed to settle, the supernatant is decanted, and the acid treatment is repeated once or twice. After thorough washing with water (twice, two liters each time), the product is suction-filtered, treated with alcohol and ether, and dried in vacuum at 300°C. Under favorable conditions, the relatively coarse, dark gray powder is obtained in 90% yield.

Kroll uses the same process with a suitable salt melt. However, instead of working in a sealed bomb, he uses an iron crucible placed in an argon atmosphere.

### B) MELTING OF Th POWDER

Small, relatively compact cylinders are formed from thorium powder under a pressure of  $6-7 \text{ tons/cm}^2$ . Crucibles of sintered thoria are suitable melt containers. The cylinder, wrapped in a tungsten wire spiral, is placed inside such a crucible which is set up in a quartz container connected to a high-vacuum pump. The material is heated with a high-frequency induction coil.

The apparatus is evacuated and the sample carefully degassed by slow heating while the vacuum is maintained. With coarse metal powder, this operation requires about half an hour, and longer with fine powders; the reaction is essentially complete when the powder reaches red heat. The temperature is then rapidly increased to melt the metal. Oxidation is prevented if air is excluded from the system until after complete cooling.

Thorium powder with a completely clean surface is noteworthy for its sensitivity to air after high-vacuum degassing at 400°C. This sensitivity is so pronounced that the metal catches fire on coming in contact with air. The material also reacts so vigorously with  $H_2$  that it becomes red hot.

III. ELECTROLYSIS

$$\begin{array}{c} \mathrm{Th}(\mathrm{NO}_3)_4 \cdot 4 \ \mathrm{H_2O} \rightarrow \mathrm{KTh}\mathrm{F_5} \rightarrow \mathrm{Th} \\ 552.2 & 366.2 & 232.1 \end{array}$$

The electrolysis of a solution of  $\text{KThF}_5$  in a NaCl-KCl melt yields very pure metal, containing only 0.02% C, 0.05% Si, 0.005% Fe and a negligible amount of other impurities.

The KThF<sub>5</sub> is prepared by dissolving 400 g. of Th(NO<sub>3</sub>)<sub>4</sub> · 4 H<sub>2</sub>O in two liters of water and adding, with constant stirring, a solution of 250 g. of KF in 400 ml. of water. The KThF<sub>5</sub> precipitate is allowed to settle, washed by decantation until the washings are free of nitrate, suction-filtered and dried for several hours at 125°C.

A cylindrical graphite crucible serves both as the electrolysis vessel and as anode. The inside diameter is 6 cm. and the height 15 cm., with a wall thickness of 1-2 cm. Current is supplied through a strip of Ni sheet wrapped around the upper part of the outside wall. The cathode is a strip of Mo sheet 0.05 mm. thick and 1 cm. wide, which reaches 2.5 cm. from the bottom. The graphite cell stands in a suitable refractory container wound with the heating filament. The entire apparatus is placed inside a sheetmetal vessel filled with thermal insulation (see also the similar arrangement described for the electrolysis of uranium).

A mixture of 250 g. each of KCl and NaCl is melted, and 30 g. of KThF<sub>5</sub> is added. When the melt is homogeneous, electrolysis proceeds, with the above Mo cathode, at a temperature of 775°C. A current of 18-20 amp. is required if the submerged cathode surface area is about 20 cm<sup>3</sup>. After 20 minutes the cathode is carefully removed from the liquid and replaced with a new piece of Mo, 30 g. of KThF<sub>5</sub> is added, and the electrolysis is continued for 20 minutes more. This procedure may be repeated several times.

For preparations on a somewhat larger scale, a larger crucible, capable of containing about 1 kg. of the melt, and a molybdenum cathode 2.5 cm. wide are used. The procedure is the same, except that the  $\text{KThF}_5$  additions are increased to 60 g., the current to 45 amp., and the cathode area to 50 cm<sup>2</sup>. Eightfold repetition of the operation in this larger equipment permits the preparation of about 130-140 g. of Th (58% yield).
The material adhering to the cathode strip is a mixture of metallic Th and solidified melt. After complete cooling, the salt and the finely powdered, readily oxidized metal fraction are removed by treatment with water. The residual coarse-grained Th is treated three times with nitric acid (1:10) and washed with water. This metal powder is then suction-filtered, washed with alcohol and ether, and dried in vacuum at  $300^{\circ}$ C.

**IV. REFINING PROCESS** 

$$ThI_4 = Th + 2I_2$$
  
739.8 232.1 507.7

The thorium metal prepared by the above process is very pure and absolutely free of oxygen. The procedure is essentially the same as that described for Ti, except that the temperature of the glowing wire is higher  $(1700 \,^\circ\text{C})$ . The starting material may be any kind of crude thorium, provided it is free of metals which will also deposit on the glowing wire; the product derived from the chloride is very suitable.

PROPERTIES:

Gray powder, solid similar to platinum. Relatively soft and ductile; these properties are unaffected by the presence of small quantities of oxide. M.p. 1830°C; d11.7. Crystal structure: type A1.

Hardly or not at all attacked by dilute acids (including hydrofluoric); dissolves readily in fuming hydrochloric acid and especially in aqua regia. Resistant to strong bases.

Thorium powder may be prepared from the solid via the hydride. The procedure is identical to that described on p. 1170 for Ti and Zr; the hydride should be decomposed above 700°C.

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### Separation of Zirconium and Hafnium

Hafnium does not occur as a separate mineral, but appears in nature as the always present companion of Zr; the Hf/Zr weight ratio is usually in the range of 0.01-0.025. The preparation of Hf or the purification of Zr thus always involves the isolation of Hf from the crude chlorination product or from commercial Zr compounds. The following fractionation processes are of practical importance:

I.	Crystallization	IV.	Ion exchange and adsorption
II.	Precipitation	v.	Partition between two solvents
III.	Distillation		

#### I. CRYSTALLIZATION

This method, which is unwieldy and applicable only to the separation of very small quantities of material, has been abandoned for all practical purposes.

#### II. PRECIPITATION

Good separation is obtained by precipitation of the phosphates; the hafnium concentrates in the less soluble fractions. A detailed description of the recovery of Hf from cyrtolite, a silicate of very high Hf content (5.5% HfO<sub>2</sub>), is given in E. M. Larsen, W. C. Fernelius and L. L. Quill in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 67.

#### III. DISTILLATION

Since the vapor pressures of  $ZrCl_4$  and  $HfCl_4$  are virtually the same, the process makes use of their adducts with  $PCl_5$  or  $POCl_3$ . When a 50-plate glass column is used, the more volatile Hf compound concentrates in the first distillation fraction (5% of the total; 2.5% to 16% HfO<sub>2</sub>); the residue remaining after distillation of 40% of the total feed contains only 0.2% HfO<sub>2</sub>.

This process seems to have recently assumed a greater industrial importance.

### IV. ION EXCHANGE AND ADSORPTION

This process, which was introduced in 1948 by Street and Seaborg for the separation of milligram quantities of Hf and Zr, becomes important in preparative work only when the small Hf fraction, and not the predominant Zr, may be retained on the column. This is achieved by selective elution of the Zr with  $1N H_2SO_4$ (which involves the formation of an anionic complex) from a column of synthetic cation exchange resin (method *a*) or by selective adsorption of Hf on silica gel from an anhydrous methanolic solution of the tetrachlorides (method *b*). Although the latter process permits larger throughputs and shorter residence times, work with anhydrous methanol involves difficulties, and the further treatment of the eluate is more troublesome. Method *b* becomes applicable chiefly in those cases when one is forced to deal with tetrachlorides, for example, when the latter are precipitated by chlorination of minerals.

a) Dowex 50 or Zeocarb 225 (350 g., with a particle size of 0.5 mm.) is treated with water for several days and then placed in a tube 120 cm. long and 2.5 cm. in diameter. A solution of 20 g. of  $ZrO(NO_3)_2 \cdot 2H_2O$  in one liter of 2N HNO<sub>3</sub> is passed very slowly through the column. (If the nitrate is not available, 24 g. of  $ZrOCl_2 \cdot 8H_2O$  is precipitated as the hydroxide, washed thoroughly, and dissolved in one liter of 2N HNO<sub>3</sub>.) The material absorbed on the column is eluted with 1N H<sub>2</sub>SO<sub>4</sub> (flow rate of 100 ml./hr.). The Hf concentrate begins to appear when 95-98% of the Zr has been recovered (passage of about nine liters of the acid); the HfO<sub>2</sub> content in the Zr salt eluted prior to this point is less than 0.01%.

[For faster throughput rates, one can use the "breakthrough method," in which the initial adsorption on the resin is omitted; the resultant separation is, however, poorer. One proceeds as follows: a solution of 2.5 g. of  $ZrO(NO_3)_2 \cdot 2H_2O$  per liter of 1N H<sub>2</sub>SO<sub>4</sub> is passed through the above column at a rate of about 200 ml./hr. Before the "breakthrough point" is reached (after the passage of about nine liters), the solution leaving the column contains mainly Zr and a small amount of Hf, whose concentration in the Zr slowly increases to 0.1%. If only seven liters is collected, the product recovered from the solution consists, for example, of 8.2 g. of oxide containing 0.047% of HfO<sub>2</sub>.]

After removal of the Zr, the Hf adsorbed on the column is eluted with stronger sulfuric acid (>1.2N); a solution of 0.05 moles of oxalic acid in one liter of  $2N H_2SO_4$  is an especially good eluent. Thus 63 mg. of Zr-free HfO<sub>2</sub> may be obtained in a 30-g. column, starting from 70 mg. of HfO<sub>2</sub> containing 8% ZrO<sub>2</sub>; the material is passed through the ion-exchange column in the form of a solution of HfOCl<sub>2</sub> · 8H<sub>2</sub>O in 675 ml. of  $1N H_2SO_4$ .

b) Silica gel (1000 g.) with a specific area of 720 m. $^{2}/g$ . is purified by treatment with nitric acid (1:1) and washing with water, activated by heating for four hours at 300 °C, and suspended in dry methanol; this suspension is placed in a tube 120 cm. long and 5.0 cm. in diameter. The column then contains about 700 ml. of methanol.

Zirconium tetrachloride (400 g., equivalent to 210 g. of  $ZrO_2$ ), with a Hf/Zr weight ratio of approximately 0.02, is dissolved in two liters of anhydrous methanol; the solution is allowed to stand for three hours and is then filtered. The filtrate advances through the column at an average rate of 20 cm./hr. (400 ml./hr.).

[The highest separation is achieved during the initial stages, as the following data illustrate:

Cumulative throughput, in g.  $ZrO_2/g$ . silica gel 0.05 0.1 0.2 0.25% Hf in Zr leaving the column0.05 0.1 0.35 0.6

Thus, one can collect an eluate containing 140 g. of  $\text{ZrO}_2$  (equivalent to 265 g. of  $\text{ZrCl}_4$ ) with a total Hf concentration of less than 0.1%.

Oddly, much better results are obtained when the operation is conducted on a larger scale. Thus a column 10 cm. in diameter and 150 cm. long charged with 8 kg. of silica gel yields, at correspondingly higher throughputs but otherwise unchanged operating conditions, 1.6 kg. of Hf-free  $ZrCl_{4}$ .]

At the point when only about 200 g. of  $ZrCl_4$  (equivalent to 100 g. of total oxide) remains in the column, the HfO<sub>2</sub> concentration in this residue becomes 10%. Further concentration is attained by elution of the column with a solution of 2.5 moles of HCl/liter of methanol (preferential desorption of the  $ZrCl_4$ ). Depending on the duration of this treatment, the final elution with 7N H<sub>2</sub>SO<sub>4</sub> yields, for example, 60% of the absorbed Hf as a 30% product, or 20% of the metal as a 60% product. These concentrates constitute a very suitable starting material for the extraction process described below.

The silica gel may be reused after reactivation.

V. PARTITION BETWEEN TWO SOLVENTS

The process is based on the preferential ether extraction of Hf from aqueous thiocyanate-containing solutions of Zr and Hf. Addition of acids or salts alters the equilibrium. Thus the presence of ether-insoluble sulfate ions shifts the distribution of Zr and Hf in favor of the aqueous phase, while the addition of acid or NH<sub>4</sub>SCN achieves the opposite effect. Since no separation can be achieved by a single-stage extraction, a multistage process must be used. The process is designed to achieve maximum separation by combining the above factors, i.e., by varying the additives in the initial and final stages. Hydrolytic reactions have also been used to advantage in this separation. The following procedure has proven effective for the processing of a raw material containing  $\sim 20\%$  HfO<sub>2</sub>.

The ether phase, which is 1N in HSCN, is prepared by shaking one liter of ether with an acidified solution of  $NH_4SCN$  (90 g. of  $NH_4SCN + 1/2$  mole of  $H_2SO_4$ ); the sulfuric acid is added in small portions during the shaking. A mixed oxide  $Zr(Hf)O_2$ , calcined at not too high a temperature, yields on evaporation with conc.  $H_2SO_4$  a product of the approximate composition 1  $ZrO_2:2SO_3$ . This product, in a concentration of 40-50 g. of oxide/liter of  $H_2O_4$ , is used as the starting material.

One liter of this freshly prepared solution (do not heat to dissolve) is treated with 600 g. of  $NH_4SCN$  and vigorously shaken for one minute with one liter of the above ether preparation. After standing for five minutes, the ether layer is siphoned off and transferred to the next stages of the process, where it is treated with solutions of the following composition:

Stage No.	2	3	4	•••
(NH <sub>4</sub> ) <sub>2</sub> SO <sub>4</sub> (g.)	80	80	80	•••
(NH <sub>4</sub> )SCN (g.)	50	25	0	
H <sub>2</sub> O (ml.)	500	500	500	

The initial extraction is repeated ten times, each time with a new batch of the ether-HSCN phase. Each of the resultant ether extracts (fractions 2-10) is then passed through all the successive stages of the process. The aqueous solutions in each stage are, of course, used over and over again; that is, the new ether fraction is treated with the thiocyanate solution remaining in that stage from the extraction of the previous ether fraction. However, it is recommended that the NH<sub>4</sub>SCN concentration in the aqueous solutions of stages 2 and 3 be gradually increased (always retaining the thiocyanate gradient shown in the table) and that thiocyanate be gradually added to the succeeding stages. The solution of the last stage must, however, always consist of 80 g. of (NH<sub>4</sub>)<sub>2</sub>SO<sub>4</sub> in 500 ml. of water, so that the ether leaving the system is always washed free of Zr and Hf. Thus, new last stages must be continually added to the series. When the thiocyanate concentration in the aqueous solution of stage 2 reaches twice the level shown in the table, this solution is "retired"; the solution from stage 3 becomes that of stage 2, the solution from stage 4 is shifted to stage 3, and so on down the line. A new thiocyanate-free aqueous stage is then added at the end of the series.

The Hf + Zr concentration and the Hf/Zr ratio must constantly be checked in each stage, since the separation depends on a large number of interdependent factors. Thus, the temperature greatly affects the partition coefficients, an effect which can be compensated for by changes in the volumetric ratios between the phases or by addition of salts.

1182

Some hydrolysis may occur in stage 1; it is recognizable by the appearance of heavy turbidity or precipitation in the aqueous layer, and may necessitate an intermediate treatment (precipitation with ammonia, followed by solution of the precipitate with  $H_2SO_4$ ).

As has been emphasized before, the process is particularly effective with partially concentrated hafnium products, as shown by the following data. Starting from 40 g. of a product with a  $HfO_2$  concentration of 18%, the aqueous layers of the various stages, after shaking with 10 liters of ether, contained the following proportions of  $HfO_2$ :

	1	2	3	4	5	6
Oxide (g.)	28	$\sim 4$	~5	2	0.6	0.2
HfO2 (%)	$\sim$ 7	$\sim 20$	> 45	> 40	> 50	>70

The corresponding figures obtained from a starting material containing 50%  $\rm HfO_2$  were:

HfO<sub>2</sub> (%) 25  $\sim 50 \sim 80 \sim 99 > 99$ 

Alternate methods: a) U.S. authors have used processes involving fractional extraction of the aqueous phase with benzene solutions of diketones [thenoyl trifluoracetone: E. H. Huffman and L. J. Beaufait, J. Amer. Chem. Soc. <u>71</u>, 3179 (1949); trifluoroacetylacetone: B. G. Schultz and E. M. Larsen, J. Amer. Chem. Soc. <u>72</u>, 3610 (1950)]. One disadvantage of the thenoyl trifluoroacetone process may be that the Zr, which is usually the major component, is preferentially extracted into the benzene phase.

b) A process in which aqueous solutions of the chlorides are countercurrently extracted with methyl isobutyl ketone in the presence of thiocyanates and thiocyanic acid has attained industrial importance [W. Fischer, H. Heitsch and G. Otto, German Patent 1,010,061, Oct. 18, 1955; Nuclear Sci. Abstr. <u>10</u>, 371 (1956)].

c) The nitrates of Zr and Hf can be selectively partitioned between aqueous nitric acid and organic solvents, particularly tributyl phosphate and ketones, the Zr being preferentially extracted into the organic phase [R. P. Cox, H. C. Peterson and G. H. Beyer, Ind. Eng. Chem. <u>50</u>, 141 (1958); Chemie f. Labor und Betrieb, August 1958, 340; W. Fischer and H. Heitsch, German Patent 1,007,306, Nov. 16, 1954; F. Hudswell and J. M. Hutcheon, Proc. Internat. Confer. Peaceful Uses of Atomic Energy, Vol. 8, 563, New York, 1956; J. Huré and R. Saint-James, ibid., 551.]

d) British authors recommend chromatographic separation on  $Al_2O_3$  or cellulose, using an organic solvent [T. V. Arden et al., Brit. Pat. 654,695, April 22, 1948, granted June 27, 1951; Chem. Zentr. 52, 4840].

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## Titanium, Zirconium and Thorium Hydrides

Ti/H

Hydrogen dissolves in the Ti metal lattice until a composition  $TiH_{0.5}$  is reached; this produces a hydride with a considerable homogeneity in the range of TiH-TiH<sub>2</sub>. The upper hydrogen concentration is attainable only with Ti and H<sub>2</sub> of the highest purity, while operating under conditions of extreme cleanliness.

The metal form best suited for the hydrogenation is Ti sponge. Titanium sheet starts to absorb  $H_2$  at 300°C and does so rapidly beginning at 400°C. Partially hydrogenated Ti reacts with carefully purified  $H_2$  even at 20°C. Hydrogen is released from highly hydrogenated products by reheating to above 400°C in high vacuum; complete desorption is achieved at 1000°C.

When it is required to absorb only a predetermined quantity of  $H_2$ , the following procedure may be employed. The metal is weighed into a boat of sintered clay (or, better, of stainless steel, provided traces of Fe in the product are not detrimental) placed in a quartz tube connected to the system with a ground joint. The apparatus consists of a glass burette with 0.1-ml. divisions provided with a leveling tube and Hg reservoir and connected to an electrolytic  $H_2$  generator; this apparatus is attached to a high-vacuum system. The quartz tube volume is measured, and the metal is degassed by heating to 550°C. Hydrides with the desired  $H_2$  content are obtained by varying the absorption temperature and the quantity of hydrogen introduced.

### Zr/H, Th/H

The preparation of Zr and Th hydrides is similar to the above procedure. Zirconium reacts very rapidly beginning at 700 °C and at atmospheric pressure is capable of dissolving 1.95 atoms of H per atom of Zr. Thorium starts to absorb  $H_2$  at 400°C; the maximum  $H_2$  concentration corresponds to a hydride composition of ThH<sub>3.24</sub>.

PROPERTIES:

Gray powder of somewhat lighter color and lower density than the parent metal powder. In contrast to the Ti and Zr hydrides, Th hydrides of high hydrogen content are labile and ignite spontaneously in air.

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### Titanium (II) Chloride, Bromide and Iodide

TiCl<sub>2</sub>, TiBr<sub>2</sub>, TiI<sub>2</sub>

 $TiCl_4 + Ti = 2 TiCl_2$ 189.7 47.9 237.6

A weighed quantity (2-3 g.) of  $\text{TiCl}_4$  is placed with the appropriate precautions in a thick-wall quartz tube, and the stoichiometric quantity of Ti filings is added. The tube is cooled in a Dry Icealcohol bath, thoroughly evacuated by means of an oil pump, and melt-sealed in such a way that its total length is 12-15 cm. It is then placed in a very slightly inclined position in a tubular electric furnace so that the Ti metal is located at the higher end and the chloride at the lower. The Ti is in the hottest part of the oven (at 800-900°C), while the section containing the TiCl<sub>4</sub> is in a cooler zone (at about 200°C). An explosion shield is recommended. If one uses a mixture such as Ti + 2 TiCl<sub>3</sub> the procedure is less dangerous but more involved.

A mixture of black and purple substances (TiCl<sub>2</sub> and TiCl<sub>3</sub>) is observed after 24 hours. As soon as all unreacted TiCl<sub>4</sub> disappears, the reactor tube is pushed deeper into the furnace, which results in a gradual disappearance of the reddish component. To achieve complete homogeneity, the mixture is heated for an

Ia.

#### P. EHRLICH

additional 4-5 days at 600-700°C. The product is black and may be dislodged from the wall by gentle tapping (the reaction with the quartz wall proceeds to only a very slight extent). The quartz tube is sawed open; the black powder is dropped into a transfer apparatus (Fig. 54, p. 75) and reheated in vacuum for 15-30 minutes at 200°C to remove the moisture absorbed during the transfer. The product is then ready for further processing.

Ib. To prepare larger quantities of  $TiCl_2$  by the same method, it is advisable to use a vertical reactor tube, in which the molten dichloride is formed on passage of  $TiCl_4$  vapor over titanium filings heated to a high temperature. A layer of Ti filings about 30 cm. high is placed on a perforated carbon plate in a fused quartz tube 110 cm. long and 4.5 cm. wide. Just underneath the carbon plate there is a graphite crucible supported by a pistonlike arrangement; this crucible collects the droplets of the product. The entire arrangement is placed in a tubular furnace; the temperature at the metal level is 1050°C, while that at the level of the collecting crucible is 900°C. After thorough flushing with Ar, gaseous  $TiCl_4$  is introduced from above in a slow stream of Ar. At the end of the reaction the graphite crucible is removed from the furnace in an atmosphere of a protective gas, and the solid block of  $TiCl_2$  is removed by gentle tapping.

### TiBr<sub>2</sub>, TiI<sub>2</sub>

These compounds are synthesized in a similar manner, except that the halogens, rather than the tetrahalides, are used as starting materials.

After weighing and before addition of the Ti filings, the  $Br_2$  must be cooled to -78 °C, since liquid  $Br_2$  and Ti react with ignition even at room temperature. This phenomenon also occurs in the sealed tube as soon as the  $Br_2$  starts to melt. The tubes, however, are capable of withstanding the pressure. With  $I_2$ , on the other hand, the conversion to tetraiodide starts only after slight heating. In both cases, further treatment is similar to that of the chloride. The quartz tube wall is attacked even less by the bromide than with the chloride, while the iodide does not react with quartz at all.

**II.**  $2 \operatorname{TiCl}_3 = \operatorname{TiCl}_2 + \operatorname{TiCl}_4$ 308.5 118.8 189.7

A high vacuum is created in a quartz tube, one end of which is filled with  $TiCl_3$  and heated to 475°C, while the other is maintained

1186

at -78 °C. The TiCl<sub>4</sub> formed via the disproportionation condenses at the cold end. Complete decomposition of 1 g. of TiCl<sub>3</sub> requires about 12 hours. When the reaction is over, the tube end containing the TiCl<sub>4</sub> is sealed off from the remainder. Since the decomposition reaction 2 TiCl<sub>2</sub> $\rightarrow$ TiCl<sub>4</sub> + Ti sets in below 475 °C, pure TiCl<sub>2</sub> cannot be obtained by this method; the product always contains 2-3% of free titanium.

On the other hand, this method may be very successfully used for the preparation of  $\text{TiBr}_2$ . At temperatures slightly above 400°C, half a gram of  $\text{TiBr}_3$  will decompose completely in 18 hours according to the equation  $2 \text{TiBr}_3 = \text{TiBr}_2 + \text{TiBr}_4$ . However, the disproportionation  $2 \text{TiBr}_2 \rightarrow \text{TiBr}_4 + \text{Ti sets}$  in above 500°C, so again free titanium may be present in the product.

III. Very pure and finely divided  $TiCl_2$  may be obtained by reduction of  $TiCl_4$  with  $H_2$  in an electrical discharge produced without electrodes.

PROPERTIES:

TiCl<sub>2</sub>: Black crystals; ignites in moist air; soluble in  $H_2O$ , evolving  $H_2$  (the same properties apply to TiBr<sub>2</sub> and TiI<sub>2</sub>). d (TiCl<sub>2</sub>) 3.13, (TiBr<sub>2</sub>) 4.31, (TiI<sub>2</sub>) 4.99.

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Titanium (III) Chloride, Bromide and Iodide

 $TiCl_3$ ,  $TiCl_3 \cdot 6 H_2O$ ;  $TiBr_3$ ,  $TiBr_3 \cdot 6 H_2O$ ;  $TiI_3$ 

TiCI<sub>3</sub>

I.

$$2 \operatorname{TiCl}_{4} + \operatorname{H}_{2} = 2 \operatorname{TiCl}_{3} + 2 \operatorname{HCl}_{379.4} 2.0 \quad 308.5 \quad 72.9$$

a) The procedure developed by Schumb et al. was modified by Klemm and Krose as follows.

The apparatus is shown in Fig. 292. Parts a, c, and d are made of fused quartz, while the container f is Pyrex. Before the start of the reaction the entire system is thoroughly dried with a stream of H<sub>2</sub>. About 25 g. of  $TiCl_4$  is then added through b, the furnace is rapidly heated to 800°C, and the cooling water for the finger d is turned on (Schumb et al. use a system made of highmelting glass and heat to  $650^{\circ}$ C only). The TiCl<sub>4</sub> in *a* is heated almost to the boiling point while a stream of H<sub>2</sub> is passed through the flask; the product is free of TiCl<sub>2</sub> only if an excess of TiCl<sub>4</sub> is present in the reaction chamber. The unreacted TiCl<sub>4</sub> is collected in container f, which is cooled with Dry Ice. After all the TiCl<sub>4</sub> has been distilled out of flask a, the current to the furnace is shut off; it should cool rapidly, since the insulation consists only of a thin asbestos layer. When the temperature drops to 120°C, the water flow to the cold finger is stopped, the finger is dried with an air stream, and the furnace is kept at 120°C for several hours in order to free the product deposited on the tip of the finger of  $TiCl_4$ . The  $TiCl_4$  left in the remaining section of the apparatus is distilled off by fanning with a flame. The system is then allowed to cool in a stream of H<sub>2</sub>, followed by a fast stream of  $CO_2$ . The container f is then disconnected at e and replaced with a transfer device (Fig. 54, p. 75); the cork stopper carrying the cold finger is pulled out from the reactor to a distance sufficient for insertion of a scraper; the TiCl<sub>3</sub> is scraped off the finger and dropped by tapping into the transfer container. The yield of the pure product is 2-3 g.



Fig. 292. Preparation of tianium (III) chloride according to Klemm and Krose. a) flask, b) charging adapter for TiCl<sub>4</sub>, c) reaction tube, d) cold finger, f) container

b) Larger quantities (150-200 g.) of less pure product (98%) can be prepared in one day in the apparatus of Fig. 293 via reduction of TiCl<sub>4</sub> with  $H_2$  on the surface of a glowing tungsten wire.

1188



Fig. 293. Preparation of titanium (III) chloride according to Sherfey. a) Pyrex reactor; e) tungsten rods, about 6 mm. in diameter; f) tungsten wire coil; m) flask for distillation of TiCl<sub>4</sub>.

The four-liter Pyrex reaction vessel a is provided with a flat-ground lid with four openings, one in the middle and the other three arranged symmetrically around it. The central 34/45 ground joint b carries a tubular adapter c closed off with a two-hole rubber stopper d through which two tungsten rods e (6 mm. in diameter) are inserted. The rods are interconnected by a tungsten wire, the thickness and length of which are determined by the available power supply. Thus, heating a wire 1 mm. in diameter and 30 cm. long to the required temperature of 1000-1100°C requires a current of 36 amp. and 8.6 v. Thinner wires should not be used, if at all possible, since they may burn out during the run; longer wires increase the process rate.

The apparatus is thoroughly flushed with pure, dry  $H_2$ , which is introduced at g and which leaves at h. When all the moisture has been removed, stopcock h is closed and i is opened, without interrupting the stream of  $H_2$ . Then TiCl<sub>4</sub> (one liter = 1700 g.) is introduced from dropping funnel k into distilling flask m and, except for a small residue, redistilled into the reaction vessel a. The distillation apparatus is then removed and the opening at n is rapidly closed off.

Only the lower third of reaction vessel a is heated with a heating mantle. The boiling TiCl<sub>4</sub> then condenses on the lid and the side walls—cooled with a fan if necessary—and, while flowing down, washes off the nascent  $TiCl_3$ . If the  $TiCl_4$  boils too violently, the solid  $TiCl_3$  particles may come in contact with the hydrogen stream, be entrained by it and plug the reflux condenser (the condenser serves only as a safety vent).

When the boiling of the  $TiCl_4$  (in the fast hydrogen stream) has reached a steady state, the tungsten wire is heated to red heat. The reduction starts immediately and is accompanied by the appearance of violet vapors of TlCl<sub>a</sub>, which condense on the walls and are largely flushed down to the bottom of the TiCl\_containing flask. Since there is a possibility that air may enter the system whenever there is a sudden cooling and resultant temporary vacuum, the H<sub>2</sub> flow rate must be carefully maintained (the air is undesirable since it may oxidize the glowing wire to the point of burnout and may also cause hydrolysis). When the  $TiCl_4$  ceases to flow unhindered along the walls of the vessel, the reaction is stopped by turning off the current to the glow wire, and the flask is allowed to cool in a stream of H<sub>2</sub>. The TiCl<sub>4</sub> may be distilled directly from the reaction beaker by replacing the lid used in the reaction with a one-hole cover. However, it is simpler to transfer the reaction mixture to a sideneck distilling flask and heat to 150°C on an oil bath. The last traces of adsorbed TiCl<sub>4</sub> are removed by heating in vacuum to 200°C; other volatile contaminants are removed at the same time. About 150 g. of TiCl<sub>3</sub>, corresponding to a yield of 10% (or 90% based on the amount of TiCl<sub>4</sub> actually consumed in the reaction), is obtained. Because it contains a small quantity of TiCl<sub>2</sub>, the product has a reducing value of 101.5%. It usually ignites even in moist air, and even more rapidly when it is still warm: transfer must therefore be carried out carefully, in an inert atmosphere.

The difficulties involved in welding on the tungsten coil may be circumvented by means of the following arrangement. Two copper tubes (diameter 6 mm., length about 30 cm.) are electrically insulated from each other and cemented in the adapter c; just above c they are provided with side fittings for connection to cooling water. The cement may be an epoxy resin such as Araldite 121 R with hardener 951.\* The upper ends of the copper tubes, which serve as bus bars, are interconnected by means of a short piece of rubber tubing; the lower ends are closed off. A strip of molybdenum sheet (0.2 mm. thick, 6 cm. long) is soldered on at the lower end of each of the two copper tubes to support the tungsten wire. A firm electrical contact between the wire (which is wound into five or six coils) and the molybdenum strips is achieved by threading the wire ends through a series of small holes in the strips, followed by bending the ends over and crimping to the strips.

II.

$$3 \operatorname{TiCl}_4 + \operatorname{Ti} = 4 \operatorname{TiCl}_3$$
  
<sup>1</sup>/<sub>10</sub>: 56.9 4.8 61.7

If pure Ti metal is available, the TiCl<sub>3</sub> may be prepared in a thick-wall pressure tube made of fused quartz or Vycor in accordance with the above equation. The procedure is essentially the same as that described for the preparation of TiCl<sub>2</sub>. A large amount of TiCl<sub>2</sub> forms initially; this stage may be recognized by the black color and moist appearance of the product, due to unreacted TiCl<sub>4</sub>. After the initial reaction the reactor tube is gradually (over several hours) pushed completely into the furnace, which is maintained at 600°C. Should a temperature gradient exist in the system, the TiCl, will sublime, in the form of violet, leaflike crystals, into the center of the pressure tube. The tip of the tube is then broken off under a blanket of protective gas; the other end, which may contain some residual unreacted Ti, is also broken off, and the TiCl, is dropped into a transfer device (cf. Part I, p. 75), in which it is heated for an additional few minutes in vacuum to 100-150 °C by fanning the vessel with a flame; the small amount of TiCl<sub>4</sub> which evolves shows that the reaction did not go to completion. The reactor walls are attacked only at the spot where the Ti metal was placed, and then only very slightly.

III.

 $3 \operatorname{TiCl}_4 + \operatorname{Sb} = 3 \operatorname{TiCl}_3 + \operatorname{SbCl}_3$ <sup>1</sup>/<sub>10</sub>: 56.9 12.2 46.3 22.8

The reduction of  $TiCl_4$  to  $TiCl_3$  with Sb does not require a complicated apparatus and may be carried out as follows:

A solution of  $SbCl_3$  (d 1.265) is reduced with Zn dust. The resultant Sb is washed several times with 0.1N HCl until free of Zn, then treated with alcohol and ether, and finally dried in a stream of  $CO_2$ . Antimony prepared by other methods does not reduce TiCl<sub>4</sub> as efficiently.

Freshly distilled  $TiCl_4$  (28 g.) is placed in a bomb tube and the Sb (6 g.) is added; the reactor tube is melt-sealed and heated for five hours at 340 °C. After cooling, both tube ends are broken off and the moist mass, in a stream of  $CO_2$ , is transferred to a three-neck flask via an adapter at neck a. Then  $CCl_4$  is added from a dropping funnel attached to neck *b* while the mixture is agitated with a stirrer inserted through the center neck; this dissolves out the unreacted  $TiCl_4$ . The mixture is allowed to settle, an adjustable siphon tube is inserted at a, and the supernatant liquid is forced out by  $CO_2$  pressure applied through b. The operation is repeated until the product is free of TiCl<sub>4</sub>. The SbCl<sub>3</sub> formed in the reaction is removed in the same manner by exhaustive extraction with ether, the last traces of which are evaporated by heating on a water bath in a stream of  $CO_2$ . The TiCl<sub>3</sub>, in the form of a violet powder, is transferred to storage under a blanket of  $CO_2$ . The yield is quantitative.

IV. Alternate method: Finely divided, very pure  $TiCl_3$  may be prepared by reduction of  $TiCl_4$  with  $H_2$  in an electric arc.

Ti**Br**3

I. The preparation of  $\text{TiBr}_3$  by method I<sub>a</sub> is similar in its essentials to that used for  $\text{TiCl}_3$ , except that the removal of  $\text{TiBr}_4$  after completion of the reaction must be carried out at a higher temperature (250 °C).

The preparation by method Ib uses the same apparatus as that for TiCl<sub>3</sub>. Since TiBr<sub>4</sub> is a solid at room temperature, the reflux condenser must be cooled with hot water or steam. The TiBr<sub>4</sub> is poured hot into the distillation flask and allowed to solidify before the apparatus is flushed with H<sub>2</sub>.

II. Sublimed  $TiBr_3$  crystals are synthesized from the elements under the same conditions as those given for  $TiCl_3$ .

TiI₃

II. Direct synthesis from stoichiometric quantities of the elements by heating in a sealed tube is similar to the preparation of  $TiCl_3$ or  $TiCl_2$  from  $Ti + TiCl_4$ . As long as the tetraiodide still accompanies the diiodide and the triiodide, the product is a solid cake which is difficult to break up by tapping. Toward the end of the reaction, after heating for several hours at 700°C, the product can be pulverized by vigorous shaking. The reaction may be completed at 180°C (reduce the temperature over a period of several days). The reaction is tested for completion by pulling out the tip of the tube from the furnace (maintained at this temperature) and cooling it with a piece of moist filter paper. The reaction is complete if after several hours only a very slight film of  $TiI_4$ is observed (the film quantity is negligible compared to the total material in the reactor).

PROPERTIES:

TiCl<sub>3</sub>: Formula weight 154.27. Violet-red to black crystals; sublimes in vacuum at 425-440 °C; decomposes to TiCl<sub>2</sub> + TiCl<sub>4</sub>

1192

above 450 °C. Readily soluble in H<sub>2</sub>O. d 2.66. Crystal structure: type D  $0_5$ .

TiBr<sub>3</sub>: Formula weight 287.65. Bluish-black crystals; decomposes to TiBr<sub>2</sub> + TiBr<sub>4</sub> at 400°C. Less soluble in H<sub>2</sub>O than TiCl<sub>3</sub>.

 $TiI_3$ : Formula weight 428.66. Violet-black needle-shaped crystals; stable up to 300°C on heating in high vacuum; decomposes to  $TiI_2 + TiI_4$  above 350°C. Dissolves slowly in H<sub>2</sub>O without evolving H<sub>2</sub>.

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TiCl<sub>3</sub> · 6 H<sub>2</sub>O

 $\begin{array}{l} TiCl_4 \rightarrow TiCl_3 \rightarrow TiCl_3 \cdot 6 \ H_2O \\ 189.7 \quad 154.3 \qquad 262.4 \end{array}$ 

Titanium (III) chloride may be prepared by cathodic reduction of TiCl<sub>4</sub> in a hydrochloric acid solution; if the concentration of TiCl<sub>3</sub> in the solution is sufficiently high, the hexahydrate precipitates on saturation with HCl. The crystallization is apparently inhibited by the presence of tetravalent titanium, and total reduction of the solution is therefore necessary.

The procedure, according to W. Fischer, is as follows.

A thick-wall cylindrical battery jar (diameter 7 cm., height 9 cm.) serves as the electrolytic cell. The center of the cell is occupied by a clay cylinder (diameter 4.5 cm., height 12 cm.). The cylinder is held in place by a cork ring with three additional holes, two for carbon anodes (placed opposite each other on the diameter) and the third for an outlet tube for the  $Cl_2$  evolved during the electrolysis. The clay cylinder is closed with a three-hole rubber stopper, which contains an inlet tube reaching almost to the bottom of the cylinder, a short outlet tube, and a lead wire to the Pt cathode. Also recommended is the insertion of an additional glass tube, used for occasional sampling of the solution as a check on the degree of reduction.

To start with,  $TiCl_4$  (19 g.) is added in drops with efficient cooling and vigorous stirring to 27 ml. of 25% HCl solution. The insoluble hydrolysis products which may be formed are filtered off through fritted glass. The clear solution is placed in the clay cylinder, and the anode chamber is filled to the same level with 25% hydrochloric acid. At 12 v. and a current density of 2.5 amp./10 cm.<sup>2</sup> of the Pt cathode, the electrolysis should require about four hours. The jar is meanwhile cooled with ice (or, if necessary, with ice-salt). Toward the end of the run, HCl is added to the solution until saturation. The hexahydrate  $TiCl_3 \cdot 6H_2O$  crystallizes best when the solution is not agitated; therefore, the introduction of HCl should be interrupted for 10 minutes every half hour. If the reduction is complete, the product crystallizes within one hour, forming a solid crystalline mass. It is placed on a coarse fritted-glass filter under CO<sub>2</sub> and the mother liquor is removed by suction. The precipitate is washed with some saturated HCl solution, followed by ether, and the crystals are dried in a vacuum desiccator over soda lime. The TiCl<sub>3</sub>.6H<sub>2</sub>O then consists of small crystals.

#### PROPERTIES:

Pale-violet, hygroscopic crystals, readily soluble in water. Decolorizes slowly by oxidation in dry air, rapidly in moist air, with formation of white  $TiO_2$  hydrate.

If a saturated solution of  $TiCl_3$  is covered with ether and saturated with HCl, a green, very unstable isomeric hexahydrate is formed [A. Stähler and H. Wirthwein, Ber. dtsch. chem. Ges. <u>38</u>, 2619 (1905)].

REFERENCE:

Private communication from W. Fischer, Hannover.

TiBr<sub>3</sub> · 6 H<sub>2</sub>O

$$\begin{array}{rcl} \text{TiBr}_{4} \rightarrow & \text{TiBr}_{3} \rightarrow & \text{TiBr}_{3} \cdot 6 \text{ H}_{2}\text{O} \\ \text{367.6} & & 287.6 & & 395.7 \end{array}$$

Titanium (III) bromide  $TiBr_3 \cdot 6H_2O$  is prepared in exactly the same way as  $TiCl_3 \cdot 6H_2O$ . A solution of 37 g. of  $TiBr_4$  in 25 ml of 34% hydrobromic acid is poured into the clay cylinder and reduced for three hours at a current intensity of 2.5 amp.

PROPERTIES:

Reddish-violet crystals. M.p. 115 °C. Soluble in  $H_2O$ , methanol, absolute alcohol and acetone; insoluble in  $CCl_4$  and benzene. Decomposes in absolute ether.

REFERENCES:

The same as for  $TiCl_3 \cdot 6H_2O$ ; see also J. C. Olsen and E. P. Ryan. J. Amer. Chem. Soc. <u>54</u>, 2215 (1932).

#### Titanium (IV) Chloride

TiCl₄

 $TiO_2 + 2C + 2Cl_2 = TiCl_4 + 2CO$ 79.9 24.0 141.8 189.7 56.0

PREPARATION

a) An intimate mixture of 30 g. of commercial  $TiO_2$  (sold under the trade name "synthetic rutile"; if natural rutile is used, it must be preground for 24 hours in a ball mill until a very fine powder is obtained), 15 g. of charcoal or carbon black, and 0.05 g. of manganese dioxide catalyst is stirred to a paste with water and 0.3 g. of soluble starch. The mixture is heated in a drying chamber with occasional stirring until a material consisting of agglomerated particles is produced; this is placed in a clay crucible, covered with a layer of carbon black, and thoroughly calcined by means of a blast burner.

The chlorination is carried out in a quartz tube (20 mm. I.D.) to which an 8-mm. quartz tube is sealed at a 45° angle. The narrower tube is inserted into a 150-ml. distilling flask, which serves as the receiver.

The larger tube is charged with the  $TiO_2 + C$  mixture and the entire apparatus is dried by fanning with a flame, while a stream

of  $CO_2$  is flowing through. The receiver is immersed in an icesalt mixture, and the reactants are slowly heated to  $450\,^{\circ}$ C in a stream of  $Cl_2$ . In the absence of a catalyst, or if the  $TiO_2$  is insufficiently ground, heating to  $1000\,^{\circ}$ C is necessary. From time to time, particularly toward the end of the run, the tube section between the furnace and the receiver is fanned with a flame to distill off any condensed  $TiCl_4$  and to prevent plugging with  $FeCl_3$ . At a flow rate of three liters of  $Cl_2$  per hour, the reaction requires 4-5 hours for completion. The receiver is removed and the product  $TiCl_4$  is freed of dissolved  $Cl_2$  (and  $COCl_2$ , if present) by drawing a stream of dry air through the tube. The yield is about 40 g.

b) A simple laboratory apparatus, which can be used for the preparation of most anhydrous chlorides (solid, liquid and gaseous), has been described by Kroll (see Fig. 294). It is made of fused quartz, which is better than porcelain since it is attacked at much higher temperature and even then produces only a single contaminant (SiCl<sub>4</sub>). The apparatus consists of a single tube in two sections, one wide and one narrow, and containing a cold finger in the large-diameter section; the latter section serves as a condensing chamber for the distillate or sublimate. The heating arrangement is divided into two parts. The heating section for the narrower tube reaches well into the wide-tube section, to prevent the chlorides from condensing in the transition section. The large-diameter section may be heated slightly if the need arises; it is only partially covered by the heating element and its exposed part may be cooled by placing it in a wooden box filled with Dry Ice; this may be necessary in the preparation of chlorides which are difficult to condense (e.g., BCl<sub>3</sub>). Chlorides which are liquid at room temperature (e.g., TiCl<sub>4</sub>) condense on the cold finger and flow into the receiver, the reactor in this case being slightly inclined. Nonvolatile chlorides deposite as solids on the finger.



Fig. 294. Preparation of anhydrous chlorides according to Kroll. The main tube and the cold finger are made of fused quartz. a) reaction chamber; b) condensing chamber. Dimensions in mm.

If the raw materials for  $TiCl_4$  are very pure  $TiO_2$  and sugar charcoal calcined in a stream of  $Cl_2$ , then the prepurification

described below becomes unnecessary. The solid charcoal may be dispensed with if the chlorine stream also contains  $CCl_4$  or  $S_2Cl_2$ .

See also the preparation of ZrCl<sub>4</sub>.

### PREPURIFICATION

The crude product is decolorized and contaminants such as  $FeCl_3$ ,  $VOCl_3$ , etc., are removed by adding 1 g. of Cu *powder* (Na amalgam or Hg may also be used) and heating the liquid to 90-100°C for 15 minutes with occasional shaking.

[If Cu *turnings* are used, the amount specified above must be increased tenfold. Oleic acid and its salts and other organic compounds, in quantities of less than 1%, are also efficient decolorizing agents [C. K. Stoddard and E. Pietz, U. S. Bur. Mines Rep. Invest. <u>4153</u>, 40 (1947)].

According to British Patent 588,657 the following purification procedure is particularly well suited for the removal of traces of vanadium. The product containing 0.072% V is mixed with 0.1% of Fe stearate and treated with H<sub>2</sub>S, causing precipitation of a black-brown sulfide; the latter is filtered off. The TiCl<sub>4</sub> then contains only 0.002% V.]

After heating with copper, the TiCl<sub>4</sub> is cooled and suction-filtered through a very dry filter funnel, and the filtrate is transferred for further purification (removal of SiCl<sub>4</sub> and dissolved nonvolatile hydrates) to the distillation apparatus described below (Fig. 295).

## A) ATMOSPHERIC PRESSURE DISTILLATION

The neck of distilling flask *a* is closed off with a ground stopper provided with a small hook, from which a thermometer is suspended on a Pt wire. The side arm passes through a condenser jacket; a small bulb b (the receiver for the forerun) and an outlet tube c filled with  $P_2O_5$  are sealed on as shown. In the initial stage of the run, the system ends in a second distilling flask e equipped with a break-seal value f (see Part I, p. 63), via which the flask is later connected to additional pieces of glassware. The apparatus is set up to point i and flasks a and e are dried by fanning with a flame while a stream of air is passed through. Then  $TiCl_4$  is placed in flask *a* and the latter is heated on an oil The forerun is collected in b, which is then sealed off, bath. while the main fraction of the material is distilled into e, which is then sealed off at d. This operation is followed by an additional distillation at atmospheric pressure: flask e is melt-sealed at ito the system shown on the right side of Fig. 295A; this part of the apparatus is very thoroughly dried by alternately evacuating the system and allowing dry air to enter. The thin-wall break-seal

value is then shattered at the prescratched point g by moving the hammer h by means of a magnet. The forerun from this distillation is collected in k and the main fraction of the distillate in receiver m, which is then melt-sealed at point l.



Fig. 295. Purification of titanium (IV) chloride by distillation: A) at atmospheric pressure; B) in vacuum.

### **B) VACUUM DISTILLATION**

Vessel m is melt-sealed at o to the rest of the apparatus shown in Fig. 295B, and the entire system is dried by fanning with a flame while a high vacuum is maintained. The TiCl<sub>4</sub> is then introduced into m and frozen with liquid nitrogen, the breakseal valve n is broken by the method described above, the system is evacuated to  $10^{-4}$  mm. and sealed at p. A forerun is collected in r by cooling this trap and gradually heating flask m; the connection to r is then sealed off. The main fraction of the product may now be distilled in one batch, that is, by collecting in cooled receiver t, sealing off at q, and repeating the vacuum distillation; or the TiCl<sub>4</sub> may be distributed into several batches and collected in traps  $s_1$ ,  $s_2$ ,  $s_3$ , etc.

#### PROPERTIES:

Colorless, acrid liquid; fumes strongly in moist air. M.p. -24.8 °C, b.p. 136 °C; d 1.73.

Hydrolyzes almost completely on solution in water; if the hydrolysis is depressed by addition of acid or if only small quantities of water are used, oxychlorides may form as intermediates. Readily forms adducts with ammonia, pyridine, nonmetal chlorides, etc.

REFERENCES:

- Preparation: a) Private communication from W. Fischer, Hannover. b) W. Kroll. Metall u. Erz <u>36</u>, 101, 125 (1939); see also A. Köster. Angew. Chem. <u>69</u>, 563 (1957).
- Prepurification: A. V. Pamfilov, A. S. Chudyakov and E. G. Standel. Zh. Prikl. Khimii 142, 232 (1935), and other papers of A. V. Pamfilov published in that period.

Distillation: K. Arii. Sci. Rep. Tohoku Imp. Univ. 22, 959 (1933); the apparatus described may be used for the distillation of other highly corrosive liquids, such as POCl<sub>3</sub>, SOCl<sub>2</sub>, etc.
Purification of TiCl<sub>4</sub> for atomic weight determinations is described by G. P. Baxter, J. Amer. Chem. Soc. <u>45</u>, 1228 (1923); <u>48</u>, 3117 (1926); E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 184.

## Ammonium Hexachlorotitanate

# $(NH_4)_2[TiCl_6]$

This is a good, easily measured starting material for preparing hydrochloric acid solutions of titanium, since it forms concentrated, stable solutions in water or dilute hydrochloric acid.

> $TiCl_4 + 2 NH_4Cl = (NH_4)_2[TiCl_6]$ 189.7 107.0 296.7

The preparation comprises precipitation of  $(NH_4)_2[TiCl_6]$  from an HCl-saturated solution, using a special apparatus which may also be employed in many other syntheses.

A 200-ml. wide-neck Erlenmeyer flask is used to hold 100 ml. of solution. The flask is closed off with a closely fitting three-hole rubber cap ("fermentation cap"). A glass stirrer, preferably of the twist drill type, is inserted in the center hole; a drop of glycerol is used for lubrication and gas seal. The use of a ground joint sealed to a mercury-seal agitator is also reccommended. Laborious centering of the stirrer is avoided and easy assembly and dismantling of the apparatus promoted by coupling the stirrer to the motor shaft (or the speed reducer shaft) by means of a piece of strong, rigid rubber vacuum hose. The direction of rotation of the stirrer is such that the center of the liquid is pushed down; higher agitation rates can be reached with this arrangement without danger of splashing, and the stirring is also more efficient.

The flask is supported at the neck by a clamp which holds it in a cooling bath at a depth so that it is covered with coolant to just below the clamp level while still leaving enough coolant underneath the flask to provide cooling of the bottom.

The gas inlet tube need not dip into the solution, since the rate of absorption of HCl in the vigorously stirred liquid is so rapid that it is almost controlled by the input rate alone; possible plugging of the inlet tube is also avoided by not letting the tube dip into solution. The HCl addition rate is controlled to avoid the formation of a mist above the stirred mixture, a point at which evaporation losses just begin. The greater the stirring rate, the higher the rate at which the HCl may be introduced, and the sooner the end of the run. Complete saturation of 100 ml. of precipitation solution requires less than one hour.

The HCl flow rate is sharply reduced toward the end of the run. The progress and termination of the HCl absorption can be followed by means of bubble counters inserted ahead of and behind the precipitation flask.

The HCl generator must be capable of yielding a continuous stream of gas and must also allow a wide range of adjustment in the flow rate; in addition, it should be easy to start, give an air-free gas stream as soon as possible after the start, and stop generating gas shortly after being turned turned off. The generator described on p. 280 fulfills these conditions less well than the apparatus developed by W. Seidel [Chem. Fabrik <u>11</u>, 408 (1938)], in which conc. hydrochloric and conc. sulfuric acids react to give a good yield of HCl; this is accomplished by dropping the acids separately onto a packing of glass beads.

If only small quantities of HCl are required, the most convenient generator is still the Kipp, which utilizes the reaction of conc. sulfuric acid with lumps of  $NH_4Cl$ , particularly since the gas does not have to be dried. However, foaming is quite pronounced at larger HCl flows.

Returning now to the precipitation of  $(NH_4)_2[TiCl]_6$ , gaseous HCl is introduced at 0°C into a solution of 6 g. of TiCl<sub>4</sub> in 100 ml. of aqueous (7:1) hydrochloric acid containing about 4 g. of NH<sub>4</sub>Cl. The HCl gas is added until saturation. Then the HCl flow is stopped, but stirring is continued until complete precipitation. If the precipitation rate is low, the yellow  $(NH_4)_2[TiCl_6]$  is obtained in the form of coarse crystals averaging 0.1 mm.

The precipitate is separated from most of the mother liquor by a short suction filtration through coarse fritted glass (without allowing air to be drawn through the compound), and the crystals are then pressed between two pieces of filter paper. If an asbestos filter is used, the compound must be repeatedly boiled with conc. hydrochloric acid and then very thoroughly washed.

PROPERTIES:

Yellow octahedra, probably of the  $K_2[PtCl_6]$  structure. May be stored for an indefinite period if moistened with hydrochloric acid and kept in a closed container; on washing with anhydrous ether and drying over conc.  $H_2SO_4$  in a vacuum desiccator, decomposes with pronounced evolution of HCl. In moist air, forms a white hydrolysis product, which is unusual in still being soluble in water.

REFERENCES:

A. Rosenheim and O. Schütte. Z. anorg. Chem. <u>26</u>, 239 (1901);
 W. Fischer and W. Seidel. Z. anorg. allg. Chem. <u>247</u>, 333 (1941);
 W. Seidel and W. Fischer. Z. anorg. allg. Chem. <u>247</u>, 367 (1941).

#### Titanium (IV) Bromide

TiBr<sub>4</sub>

I.	$TiCl_4 \ +$	4  HBr =	TiBr <sub>4</sub> +	4 HCl
•	189.7	323.7	367.6	145.8

Due to the long time required (30 hours), the original method described by Thorpe in 1856 (bubbling of HBr through warm  $TiCl_4$  until the boiling point of the solution equals that of  $TiBr_4$ ) has been modified as follows.

Receiver b of the apparatus shown in Fig. 296 is cooled with liquid nitrogen or Dry Ice, and pure, dried HBr is condensed in until enough liquid is present. The section above d is then broken off, TiCl<sub>4</sub> is added to a, and the apparatus is resealed at e. Stopcock f is closed, i is opened, and TiCl<sub>4</sub> is slowly distilled into container b, which is cooled with Dry Ice; the initial reaction is quite violent. By periodically removing the coolant, it is possible to bring the reaction to completion. The gas mixture evolved (essentially HCl) is vented through stopcock i, while the HBr is condensed in c by proper cooling. The mixture is allowed to warm up to room temperature in order to accelerate the reaction, and the condensation of fresh HBr into the flask is repeated several times. Finally, vessel b, which contains an already very pure crude product, is sealed off at g; if desired, flask c may be sealed at point h to a distillation apparatus (such as the one described

in the preparation of  $TiI_4$ ) and the  $TiBr_4$  further purified by vacuum distillation.



Fig. 296. Preparation of titanium (IV) bromide.

II.

 $TiO_2 + 2C + 2Br_2 = TiBr_4 + 2CO$ 79.9 24.0 319.7 367.6 56.0

The already-described method of preparation of  $TiCl_4$  is modified only to the extent that the stream of  $Cl_2$  is replaced by dry  $CO_2$  which passes through a 60°C wash bottle containing 135 g. of  $Br_2$ ; the bromine-saturated  $CO_2$  then passes over the reaction mixture (30 g. of  $TiO_2 + 15$  g. of wood charcoal), which is heated to about 600°C. A mixture of  $TiBr_4$ ,  $CBr_4$  and free  $Br_2$ collects in the receiver. The last two products are distilled off in a stream of pure  $CO_2$  bubbled through the melt, leaving the  $TiBr_4$ as the residue. Cooling to room temperature produces a solid mass, which may be purified by multiple distillation. The yield is 80%.

III.

 $Ti + 2 Br_2 = TiBr_4$ 47.9 319.7 367.6

If metallic Ti is available, the compound may be easily synthesized from the elements (see the procedure for the preparation of titanium dihalides). A weighed amount (5-6 g.) of freshly distilled  $Br_2$  is placed in a thick-wall quartz tube cooled with Dry Ice, crude Ti is added (somewhat more than the stoichiometric quantity), and the tube is sealed under high vacuum. The  $Br_2$ begins to melt on removal of the coolant; the reaction starts immediately and flames appear. After completion of the reaction the tube is opened and the TiBr<sub>4</sub> is distilled off; it may be purified by multiple distillation (see TiI<sub>4</sub>). PROPERTIES:

Amber yellow, octahedral crystals. M.p.  $40^{\circ}$ C, b.p.  $230^{\circ}$ C; d 3.25. Extremely hygroscopic, absorbs moisture with hydrolytic decomposition. Very readily soluble in alcohol, moderately in ether; soluble in 34% hydrobromic acid and in conc. hydrochloric acid. Crystal structure: type D1<sub>1</sub>.

REFERENCES:

- I. W. Biltz and E. Keunecke. Z. anorg. allg. Chem. <u>147</u>, 171 (1925); W. Klemm, W. Tilk and S. von Müllenheim. Z. anorg. Chem. <u>176</u>, 1 (1928).
- II. See also J. C. Olsen and E. P. Ryan. J. Amer. Chem. Soc. <u>54</u>, 2215 (1932), as well as R. C. Young in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 114.
- III. See also J. M. Blocher Jr., R. F. Rolsten and I. E. Campbell. J. Electrochem. Soc. <u>104</u>, 553 (1957).

Purification of TiBr<sub>4</sub> for atomic weight determination is described by G. P. Baxter and A. Q. Butler, J. Amer. Chem. Soc. 50, 408 (1928); E. H. Archibald, The Preparation of Pure Inorganic Substances, New York, 1932, p. 185.

# Zirconium (IV), Hafnium (IV) and Thorium (IV) Chlorides and Bromides

ZrCl<sub>4</sub>, HfCl<sub>4</sub>, ThCl<sub>4</sub>; ZrBr<sub>4</sub>, HfBr<sub>4</sub>, ThBr<sub>4</sub>

**ZrCI**₄

$\mathrm{Zr}\mathrm{O}_2$	+	2 C	+	$2 \operatorname{Cl}_2$	=	$\mathrm{Zr}\mathrm{Cl}_4$	+	$2\mathrm{CO}$
123.2		24.0		141.8		233.0		56.0

An intimate mixture of one part of pure  $ZrO_2$  and two parts of calcined carbon black or sugar charcoal is placed in a porcelain boat and heated at 500 °C in a stream of  $Cl_2$ ; or, preferably,  $ZrO_2$  with no admixtures is chlorinated in a  $Cl_2$ - $CCl_4$  gas mixture produced by passing  $Cl_2$  through a wash bottle (70°C) filled with  $CCl_4$ . The initial chlorination temperature is 350°C, but is gradually raised to 700°C.

The equipment is similar to that described on p. 889 for the preparation of BeCl<sub>2</sub>, except that, when working with  $Cl_2-CCl_4$ , a trap for the unreacted  $CCl_4$  must be inserted in line after tube A. Since  $ZrCl_4$ , which sublimes at 331°C, is difficult to recondense, it is advisable to use a long tube (600 mm.), preferably of Vycor. The additional resublimation at 300-350°C is carried out in a stream of  $H_2$ , a treatment which more effectively removes the oxide and FeCl<sub>3</sub> present.

Alternate method: The industrial chlorination of ZrC prepared from  $ZrSiO_4$  is described by W. J. Kroll et al. [Trans. Electrochem. Soc. <u>89</u>, 263 (1946); <u>92</u>, 187 (1947); J. Electrochem. Soc. <u>94</u>, 1 (1948)].

PROPERTIES:

White crystalline powder. Sublimation point  $331^{\circ}$ C, m.p. (under pressure)  $438^{\circ}$ C; d 2.80. Yields a mist of hydrochloric acid in moist air; violently decomposed by H<sub>2</sub>O, forming the oxychloride. Soluble in alcohol and ether. Crystal structure: type D 1<sub>1</sub>.

### HfCl<sub>4</sub>, ThCl<sub>4</sub>

The same general method is used for  $HfCl_4$  and  $ThCl_4$ ; in the case of  $ThCl_4$ , the  $Cl_2$ - $CCl_4$  mixture should be replaced with  $Cl_2$ - $SCl_2$ , since this allows reducing the temperature to 700°C instead of 900°C.

### ZrBr<sub>4</sub>, HfBr<sub>4</sub>, ThBr<sub>4</sub>

The preparation of the bromides in a  $Br_2$ -saturated nitrogen stream requires high temperatures if practical reaction rates are to be achieved. The oxide-carbon mixture must usually be heated to about 1100°C; this temperature is easily attained with a gas-air blast burner provided the quartz reactor is embedded in porous, refractory gravel ("Diatomite" gravel).

The preparation of  $HfBr_4$  and the lower bromides  $HfBr_3$  and  $HfBr_2$  is described in W. C. Schumb and C. K. Morehouse, J. Amer. Chem. Soc. <u>69</u>, 2696 (1947).

REFERENCES;

D. Lely and L. Hamburger. Z. anorg. Chem. <u>87</u>, 209 (1914);
A. Voigt and W. Biltz. Z. anorg. allg. Chem. <u>133</u>, 277 (1924);
O. Hönigschmid, E. Zintl and F. González. Z. anorg. allg. Chem. <u>139</u>, 293 (1924); J. H. deBoer and J. D. Fast. Z. anorg. allg. Chem. <u>187</u>, 177 (1930); W. Fischer, R. Gewehr and H. Wingchen. Z. anorg. allg. Chem. <u>242</u>, 161 (1939); J. P. Coughlin and E. G. King. J. Amer. Chem. Soc. <u>72</u>, 2262 (1950); for the bromides, see also R. C. Young and H. G. Fletcher in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London, 1939, pp. 49, 51.

# Thorium Chloride

## ThCl₄ · 8 H₂O

A solution of thorium hydroxide in excess hydrochloric acid is evaporated until sirupy and is then allowed to cool and crystallize. Further purification, in particular, removal of Fe and SiO<sub>2</sub>, is best achieved by shaking with an ether-aqueous hydrochloric acid mixture. The experimental arrangement is the same as described for the preparation of  $(NH_4)_2[TiCl_6]$ , p. 1199.

The crystals are dissolved in the minimum quantity of 6N HCl, filtered through asbestos, and shaken twice with ether to remove the iron. Silicic acid precipitates during the evaporation and is also filtered off. The filtrate is cooled to  $0^{\circ}$ C, and HCl gas is passed through until saturation. An equal volume of ether is added and the mixture is treated with additional HCl until homogeneous. Pure white crystals of ThCl<sub>4</sub>  $\cdot 8 H_2$ O crystallize; these are filtered, washed with ether, and dried.

PROPERTIES:

Formula weight 518.08. Deliquescent in moist air, readily soluble in water and alcohol. Soluble in ethylenediamine.

**REFERENCES**:

C. B. Kremer. J. Amer. Chem. Soc. <u>64</u>, 1009 (1942); T. Muniyappan. Master's Dissertation, University of Illinois, 1955.

# Titanium (IV), Zirconium (IV) and Thorium (IV) lodides

## $TiI_4$ ( $ZrI_4$ , $ThI_4$ )

Syntheses I and II (described below) start from crude Ti (prepared from TiCl<sub>4</sub> and Na), which is allowed to react with  $I_2$  vapor, while in method III a commercial fine Ti-Al alloy powder (Altam 70%, i.e., containing 70% Ti) is boiled in a solution of  $I_2$  in CS<sub>2</sub>. Upon removal of the solvent, the AII<sub>3</sub> is bound in a non-volatile complex KAII<sub>4</sub>, while the TiI<sub>4</sub> is distilled off. This method is recommended for larger-scale preparations.

Ti	$+ 2I_{2}$	=	TiI₄
47.9	507.7		555.6

I. Crude Ti (20 g.) is treated with dilute hydrofluoric acid, washed with distilled water and alcohol, and dried. It is then placed in the center bulb e of the apparatus shown in Fig. 297, which is sealed off at f as close to the bulb as possible. The apparatus is made of high-melting glass, preferably fused quartz ware if highest purity is to be obtained. Filling tube a is closed off with a rubber stopper; the system is evacuated and dried by fanning with a flame while vacuum is maintained. The stopper at a



Fig. 297. Preparation of titanium (IV) iodide according to the method of Blocher and Campbell.

is removed for a while, and doubly sublimed, carefully dried  $I_2$  (100 g.) is added via tube a. The  $I_2$  is melted and transferred as a melt into bulb c, which is then sealed off at b. Bulb c is cooled with Dry Ice; the system is evacuated to  $10^{-3}$  mm. and sealed at i. The center bulb is heated to  $525^{\circ}$ C, and the two side bulbs are alternately heated and cooled with air, to produce a slow stream of  $I_2$  vapor which flows back and forth over the heated metal. The reaction is complete after three passes. The conversion is quantitative, based on the metal content of the Ti. Nonmetallic impurities are left as a residue in e. If the compound is to be resublimed or subdivided into portions, additional bulbs are fused onto h as described in the preparation of BeCl<sub>2</sub> (p. 889).

II. If a particularly pure product is desired, one may proceed as follows: 2 g. of Ti powder is placed in section a of the Pyrex apparatus shown in Fig. 298 and heated for one hour at 500°C in

high vacuum (provided by a pump attached at o). The material is then cooled to room temperature, and thin glass partition d is broken by means of a magnet and steel ball g (which is then removed from the system by sealing off at h; bulb b, which contains 10 g. of I<sub>2</sub>, is thus connected to the rest of the apparatus. The latter is now sealed off at i and the pump is turned off. The I<sub>2</sub> vapor reacts immediately (sometimes slight heating is necessary) with the Ti to give a quantitative yield of  $TiI_4$  (in the preparation of  $ZrI_4$ , it is necessary to heat the apparatus for several hours in an electric furnace at 200°C). When the reaction is complete, the apparatus is sealed off at m. After breaking partition f, the gases liberated during the reaction are removed by means of a high-vacuum pump connected at q. The system is resealed at k, the TiI<sub>4</sub> is sublimed from ainto c by heating the former, and construction n is sealed off. The pump may now be connected at p, e broken, and any gas



Fig. 298. Preparation of titanium (IV) iodide according to Fast: d, e and f are break-seal values; g is a steel ball.

evolved during the sublimation removed. The compound may then be removed, as desired, through p. If the TiI<sub>4</sub> is to be stored, the system is sealed off at l.

The preparation of  $ZrI_4$  or  $ThI_4$  is similar.

III. In the method of Blumenthal and Smith, the apparatus (Fig. 299) consists of a two-liter, long-neck, round-bottom flask a, two smaller round-bottom flasks b and c (500 ml., 250 ml. respectively), a condenser and a receiving flask d (250 ml.). The multihole rubber stopper in the large flask carries the following: 1) an annular heating device consisting of a glass tube g which terminates at the bottom in a closed sphere; steam is introduced via a thin inner rubber tube which reaches down to the sphere; 2) a dropping funnel; 3) a reflux condenser; and 4) a glass tube with a largerdiameter filter section (the latter is in flask a and is filled with glass wool). All openings to the atmosphere are protected with drying tubes filled with silica gel. Before the start of the preparation, 10 g. of KI is placed in flask b. The apparatus is assembled and dried by fanning with a flame while a stream of air is drawn through. The rubber stopper is raised rapidly and 127 g. of I<sub>2</sub>, dissolved in 600 ml. of CS<sub>2</sub>, and 50 g. of finely powdered Altam 70% alloy (equivalent to 1/2 mole of free Ti, the remainder being the oxide) is added to flask a. The solution is brought to a boil by passing steam through the heating device. The heating is



Fig. 299. Preparation of titanium (IV) iodide according to Blumenthal and Smith: e rubber connection with pinchcock; g steam heater consisting of an outer glass tube with a sphere and an inner rubber tube; h filter with glass wool packing.

continued for one hour with occasional shaking, resulting in quantitative formation of the iodides (TiI<sub>4</sub> and AlI<sub>3</sub>). These are soluble in  $CS_2$ .

The mixture is cooled to room temperature and compressed dry air is introduced via the reflux condenser, forcing the solution through the glass wool filter and into flask b. This transfer is carried out in stages, since the iodide solution in flask b must be concentrated from time to time by distilling excess CS<sub>2</sub> into flask d. Finally, three 100-ml. portions of CS<sub>2</sub> are added to rinse out the last traces of product in a; these are also transferred to b.

The rubber connection e is closed off with a clamp and flask a is removed from the system. Flasks b and c are heated in a water bath to 80 °C until all the CS<sub>2</sub> distills into d. A slow stream of dry N<sub>2</sub> is introduced through stopcock f, and flask b is strongly heated with a burner while flask c is cooled with cold water; this causes the AlI<sub>3</sub> to react quantitatively with the KI to form nonvolatile KAlI<sub>4</sub>; the TiI<sub>4</sub> meanwhile distills into c. The distillation is ended when colored vapors can no longer be observed.

The crude product (90% yield) contains 95.1% TiI<sub>4</sub>, 4.6% free I<sub>2</sub> and 0.3% iodides of other metals. Since 98% of the CS<sub>2</sub> is recycled, and since KI and I<sub>2</sub> may be recovered from the KAII<sub>4</sub> melt by air oxidation:

$$4 \text{ KAlI}_4 + 3 \text{ O}_2 = 4 \text{ KI} + \text{Al}_2 \text{ O}_3 + 6 \text{ I}_2,$$

the process is suited for the preparation of large quantities of Til<sub>4</sub>.

IV. Alternate method: If metallic Ti is not available,  $TiI_4$  may be prepared by the method of Hautefeuille (1867). The procedure is similar to method I for the preparation of  $TiBr_4$ .

The tetraiodides of Ti, Zr and Th may be produced from the oxides with the aid of  $All_3$ .

PROPERTIES:

Red-brown octahedra crystallizing in type D  $1_1$ , but transformed on prolonged storage to a modification with a lower degree of symmetry. M.p. 150°C, b.p. 377°C; d 4.40. Fumes strongly in air; dissolves rapidly in water with hydrolytic decomposition.

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# Titanium (III) Oxychloride

## Ti**OC**l

 $TiO_2 + 2 TiCl_3 = 2 TiOCl + TiCl_4$ 79.9 308.5 198.7 189.7

A quartz tube is thoroughly baked while under high vacuum. It is then charged (under a nitrogen blanket) with TiCl<sub>3</sub> (50% excess) and TiO<sub>2</sub>. The tube is evacuated  $(10^{-5} \text{ mm.})$ , sealed off and placed in a furnace with a temperature gradient so that one third of the tube, containing the  $TiO_{a}$ - $TiCl_{a}$  mixture, is at 650°C while the remainder is at 550°C. The reaction ends in about 12 hours; the excess TiCl, and a small amount of yellowish-brown crystals of TiOCl pass into the cold zone. The hot zone contains a brown, finely crystalline cake of TiOC1. If heating in the temperature gradient is continued for several days all of the TiOC1 migrates to the colder zone and deposits as beautiful long crystals. The TiOC1 is isolated by distilling the TiCl4 into the empty half of the tube, freezing it there, and cutting the tube in two. The mixed crystals of TiOCl and TiCl<sub>3</sub> are then treated with dimethylformamide, in which TiCl, dissolves readily, forming a blue solution. The TiOCl residue is repeatedly washed with dimethlyformamide, followed by alcohol and ether, and dried in vacuum.

The compound may also be prepared by a similar procedure via the reaction of TiCl<sub>3</sub> with Fe<sub>2</sub>O<sub>3</sub>, SiO<sub>2</sub>, H<sub>2</sub>O or O<sub>2</sub>.

PROPERTIES:

Golden-yellow to red-brown crystals; decomposes slowly in air. Decomposes to  $TiO_2$  and  $TiCl_4$  on heating in an open annealing tube.

REFERENCE:

H. Schäfer, F. Wartenpfuhl and W. Weise, Z. anorg. allg. Chem. <u>295</u>, 268 (1958).

# Titanium (IV) Oxychloride

TiOCl<sub>2</sub>

I.

An excess of  $TiCl_4$  is treated with  $As_2O_3$ , resulting in a highly exothermic reaction which goes to completion if caking of the

solid product is avoided. The yellowish substance obtained is freed of  $AsCl_3$  and excess  $TiCl_4$  by suction filtration in the absence of air and thorough washing with absolute pentane or  $CCl_4$ . Residual solvent is removed by vacuum distillation at room temperature. The product contains traces of arsenic.

II. 
$$TiCl_4 + Cl_2O = TiOCl_2 + 2Cl_2$$
  
189.7 86.9 134.8 141.8

A stream of  $Cl_2O$  diluted with dry  $O_2$  is introduced through a large-diameter inlet tube into a two-neck 250-ml. flask containing about 100 ml. of TiCl<sub>4</sub> (the TiCl<sub>4</sub> is distilled into the flask under conditions of complete exclusion of moisture). The gas is prepared by passing a stream of  $O_2$ -Cl<sub>2</sub>, predried with  $P_2O_5$ , over HgO. The latter is contained in a glass tube provided with a cooling jacket and able to rotate (Liebig condenser).

Plugging of the inlet tube with solid TiOCl<sub>2</sub> is prevented by sealing a glass spatula to the bottom of the flask in such a way that it projects a few centimeters into the tube. Occasional rotation of the flask around the inlet tube then keeps the latter free.

The  $O_2$ -Cl<sub>2</sub>O mixture is bubbled in until the formation of a crystalline paste makes this impossible. The mixture is allowed to stand overnight, whereupon any small quantity of hypochlorite still present decomposes to Cl<sub>2</sub> and additional TiOCl<sub>2</sub>. The product is filtered in the absence of moisture and washed several times with high-purity CCl<sub>4</sub> which has been distilled over P<sub>2</sub>O<sub>5</sub>; the product is freed of the CCl<sub>4</sub> by evaporating the latter in a stream of a dry gas, and is then kept in vacuum for a short time.

The yield is practically quantitative, based on  $Cl_3O$ ; based on TiCl<sub>4</sub> actually used, it is about 50%.

PROPERTIES:

Pale yellow, hygroscopic, crystalline powder. Sparingly soluble in CCl<sub>4</sub>, benzene and similar solvents, moderately soluble in ethyl acetate, readily soluble in ethereal hydrochloric acid (decomposition). Hydrolyzes in moist air, giving a white color. Dissociation to TiCl<sub>4</sub> and TiO<sub>2</sub> begins at 180°C. d 2.45.

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II. K. Dehnicke. Angew. Chem. <u>75</u>, 417 (1963); Angew. Chem. (International Ed. in English) <u>9</u>, 325 (1963).

> Zirconium Oxychloride ZrOCl<sub>2</sub> · 8 H<sub>2</sub>O

The anhydrous compound is unknown. Of the existing hydrates, with 2, 3, 5, 6 and 8 moles of  $H_2O$ , the last is the most important,

1210

since it crystallizes as a sparingly soluble compound from aqueous solutions containing HCl. In a solution containing about 1.2 g. of  $ZrOCl_2 \cdot 8H_2O$  per 100 ml. of  $H_2O$ , the flat minimum section of the solubility curve corresponds to a concentration of 7-8 moles of HCl/liter at 0°C. The octahydrate is readily recrystallized and can therefore be prepared in very pure form.

PREPARATION

$$ZrO_2 \rightarrow ZrOCl_2 \cdot 8 H_2O$$
123.2 322.3

I. Since zircon  $ZrSiO_4$ , a mineral found in nature, is more difficult to work with, it is better to start from zirconia  $ZrO_2$  (baddeleyite), which is calcined, finely ground (the coarser particles are screened off with silk gauze), and converted to the sulfate by evaporation or treatment for several days with an excess of warm conc.  $H_2SO_4$ . The solid residue, which consists of  $Zr(SO_4)_2$  and unreacted  $ZrO_2$ , is taken up in water (the solid is added in small portions to prevent heating of the solution). The sulfate dissolves slowly, and its solution may be aided by acidifying the water with some hydrochloric acid. The resultant milky suspension, which contains solid undissolved  $ZrO_2$  and SiO<sub>2</sub> (or  $ZrSiO_4$ ), is allowed to stand for several hours and filtered.

The weakly acidic sulfuric acid solution is precipitated with ammonia and the hydroxide is filtered off. If the precipitate still exhibits a high Si content, it is dissolved in conc. hydrochloric acid and the solution is evaporated to dryness; this procedure is repeated several times. On redissolving in water,  $SiO_2$  and some basic zirconium chloride become the insoluble residue. If no Si is evident in the hydroxide, the fresh gel is dissolved in cold hydrochloric acid and the oxychloride is allowed to crystallize by adding conc. hydrochloric acid or saturating with HC1. The crystals are filtered and washed with 8N HC1.

II. When the starting material is high in  $SiO_2$  and, in general, if a platinum dish is available, the  $ZrO_2$  may be evaporated with a mixture of conc.  $H_2SO_4$  and 40% hydrofluoric acid instead of with pure  $H_2SO_4$ . The temperature required for this procedure is lower than in the preceding method. The subsequent steps are as described in method I.

III. The octahydrate  $ZrOCl_2 \cdot 8H_2O$  may also be prepared as follows. A suspension of freshly precipitated zirconium hydroxide in  $H_2O$  is dissolved in cold dilute hydrochloric acid; after filtering, the  $ZrOCl_2 \cdot 8H_2O$  is crystallized by evaporation (if necessary, by

adding conc. hydrochloric acid). The starting zirconium hydroxide is prepared by precipitating a solution of  $K_2 Zr F_6$  with ammonia; the precipitate, which contains a basic fluoride, must be treated for a short time with conc.  $H_2SO_4$  (to remove the HF) and redissolved in  $H_2SO_4$ . The pure hydroxide is precipitated with ammonia.

IV.  $ZrCl_4$  is dissolved in water (do not heat to dissolve—if necessary, add some hydrochloric acid); the solution is filtered and the oxychloride is precipitated by making the solution 7-8N in HCl.

The crude chlorination products of those zirconium-containing minerals that are difficult to break down must be rechlorinated with  $Cl_2$  at 1000°C, yielding crude chlorides, which can then be purified via method IV.

PREPARATION BY RECRYSTALLIZATION OF THE OXYCHLORIDE

V. The fact that  $ZrOCl_2 \cdot 8H_2O$  dissolves readily in water and is insoluble in 7-8N (25-30%) hydrochloric acid allows this compound to be used as an intermediate in the purification of Zr salts. Although complete isolation of zirconium cannot be achieved, this method eliminates not only Al, Fe, Nb, Ta, the rare earths and many other elements, but also Ti, the removal of which otherwise involves great difficulties. Thus, for example, the Al content may be reduced from 0.035% to 0.0015% by only one recrystallization; the decrease in Fe content is of the same order of magnitude. Reprecipitation of the oxychloride is thus more effective than that of the sulfate, described on p. 1232.

Since the molar solubility of  $HfOCl_2 \cdot 8H_2O$  is identical to that of the Zr salt, the Hf/Zr ratio remains unchanged.

The strongly acidic HCl solution of  $ZrOCl_2 \cdot 8H_2O$  is evaporated on a water bath until crystallization is incipient and is then treated with an equal volume of conc. hydrochloric acid; the mixture is heated (do not allow too much HCl to escape) and, if necessary, 25% hydrochloric acid is added to the warm mixture until solution is complete and the mixture contains, at most, 39 g. of oxychloride, i.e., 15 g. of  $ZrO_2$  per 100 ml. The solution is mechanically stirred and its temperature is allowed to drop to a point where it still feels warm to the hand; it is then cooled with ice. After stirring for 30 minutes at 0°C, the product is filtered through a mediumporosity fritted glass and washed with 25% hydrochloric acid precooled to 0°C.

The filtrate still contains about 1.5 g. of the oxychloride (or 0.6 g. of  $ZrO_2$ ) per 100 ml.

VI. A simpler method gives a less complete precipitation. One proceeds as follows.

First 25 g. of  $ZrOCl_2 \cdot 8H_2O$  is dissolved in a mixture of 6 ml. of conc. hydrochloric acid and 100 ml. of  $H_2O$ . The solution is heated to 70°C and filtered. The filtrate is concentrated to 75 ml. and allowed to cool without stirring. The crystallizing salt is suction-filtered on a fritted glass and washed with a cold 1:1 alcohol-conc. hydrochloric acid mixture, in which the oxychloride is very sparingly soluble. The yield is 10 g. of purified material; an additional 7 g. may be recovered from the mother liquor by further evaporation and crystallization.

SYNONYM:

Zirconyl chloride.

PROPERTIES:

Tetragonal prisms or needles. Deliquescent in moist air, evolves HCl and becomes dull in dry air. Soluble in  $H_2O$  (slight hydrolysis) and alcohol. Lower hydrates are formed on heating in a stream of HCl. Liberates HCl on heating in air and the solubility in water is gradually lost; reverts to the oxide on calcination.

Precipitation of an alcoholic solution with ether or acetone yields dizirconyl chloride  $Zr_2O_3Cl_2 \cdot 5H_2O$ , which is sparingly soluble in water. The same compound deposits when a dilute aqueous solution of zirconyl chloride is allowed to stand for a month.

## Hafnium Oxychloride

The preparation and properties of  $HfOCl_2 \cdot 8H_2O$  are virtually identical to those of the Zr compound.

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VI. W. C. Schumb and E. Pittman in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 121.

### Lower Titanium Oxides

TiO, Ti<sub>2</sub>O<sub>3</sub>

I. The surest preparation of defined lower Ti oxides involves sintering with metallic Ti.

 $Ti + 3 TiO_2 = 2 Ti_2O_3$ ;  $Ti + TiO_2 = 2 TiO_47.9$  239 7 287.6 47.9 79.9 127.8

Filings are prepared from a Ti sheet and ground to pinhead size; a magnet is used to free them from the Fe picked up during the machining operation. The filings are etched with dilute hydrofluoric acid, rinsed with acetone, and rapidly dried. They are mixed with TiO<sub>2</sub> in proper amounts and the mixture is pressed into tablets or rods; these are heated in high vacuum to  $1600^{\circ}$ C in the arrangement illustrated in Fig. 300.



Fig. 300. Synthesis of lower titanium oxides (dimensions in mm.).

Two Tammann crucibles (10 mm. and 14 mm. I.D.), made of sintered clay, are placed one inside the other, and the assembly is placed in a tube (20 mm, I.D. and 400 mm, long) made of the same material and closed at one end. The 14-mm. crucible is loosely covered by the closed end of an identical crucible, as shown. The outer (20 mm. I.D.) tube is connected to a high-vacuum system via a ground joint cemented on with picein. This arrangement of three concentric tubes is needed because the outer corundum tube is not vacuum-tight at the reaction temperature of 1600°C. The gas used for flushing the annular space between the surrounding graphite heater and the inner corundum tube is hydrogen, which diffuses inside. Thus, in new tubes, the inside pressure rises from a satisfactory high vacuum to 1 mm, within 10 minutes at 1600°C; the pressure rises even more rapidly in older tubes. The pump must therefore be left on during the entire heating period. The above arrangement of crucibles prevents the hydrogen from diffusing to the reactants before it can be removed by the vacuum pump.

The mixture is heated at 1600°C for 15 minutes in a Tammann furnace; this is obviously insufficient to bring about complete reaction. The Ti is not completely consumed, but it becomes so brittle that it is readily pulverized in an agate mortar. This fine powder is reheated and the product is then homogeneous.

If Ti powder is used as the starting material, a single but longer heating run is sufficient (1/2 hour at 1600 °C).

Materials with a low oxygen content are best subjected to a preliminary homogenization treatment, either by high-frequency heating in high vacuum, or by so-called button-melting in an electric arc, which is familiar in titanium metallurgy.

The above procedure is generally applicable and may be used for the preparation of lower oxides of other elements closely related to titanium, e.g., Zr, Hf, V, Nb, etc.

In many cases it has proved more convenient not to start each run from the metal; in those cases, a larger quantity of the lowoxygen compound is prepared and then used as a stock raw material.

II. Reduction of TiO<sub>2</sub> in a stream of H<sub>2</sub> at 1250°C yields a product of composition TiO<sub>1.6</sub>, at 1430°C and longer reaction times up to TiO<sub>1.46</sub>. At 1000°C, the reduction of TiO<sub>2</sub> in a TiCl<sub>4</sub>-saturated stream of H<sub>2</sub> also yields a small amount of violet-colored Ti<sub>2</sub>O<sub>3</sub>, besides the other products.

III. When  $TiO_2$  is reduced with carbon, the formation of mixed TiO-TiC crystals cannot be entirely prevented. According to Shomate, heating in vacuum at 1400°C for 20 hours yields  $Ti_2O_3$  via the reaction 2  $TiO_2+C=2$   $Ti_2O_3+CO$ , in agreement with the observations of Junker, who found that significant amounts of carbide are formed only above 1600°C.

PROPERTIES:

TiO: Formula weight 127.80. Golden yellow powder. M.p. 1750°C; d 4.89. Crystal structure: type B1 (NaCl type). The rock salt phase is homogeneous over a wide range of compositions  $(TiO_{1,3}-TiO_{0,6})$ . TiO dissolves in dilute hydrochloric and sulfuric acids with partial oxidation:  $Ti^{2+} + H^+ = Ti^{3+} + \frac{1}{2} H_2$ .

Ti<sub>2</sub>O<sub>3</sub>: Formula weight 143.80. Dark violet powder. M.p. ~1900°C; d 4.49. Crystal structure: type D5<sub>1</sub>.

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# Titanium (IV) Oxide

### TiO<sub>2</sub>

Titanium (IV) oxide crystallizes in three modifications of decreasing stability: rutile, anatase and brookite. Whether the synthesis of brookite has been achieved is still in doubt. Anatase is formed via the hydrolysis of Ti halides at not too high a temperature ( $600^{\circ}$ C) or via low-temperature calcining ( $\sim 700^{\circ}$ C) of precipitated titanic acid. The lattice is stabilized by adsorbed anions, among which the most effective are sulfate and phosphate. Pure TiO<sub>2</sub> calcined at high temperature always yields the rutile lattice.

I.

$$\begin{array}{rl} TiCl_4 \ \rightarrow \ TiO_2 \\ 189.7 & 79.9 \end{array}$$

Very pure TiO<sub>2</sub> is readily prepared by hydrolysis of prepurified and repeatedly distilled TiCl<sub>4</sub>. The chloride is hydrolyzed in Pyrex vessels cooled in ice and the residual titanic acid is precipitated by addition of ammonia. The mixture is boiled for one hour, filtered and thoroughly washed until free of chloride (if necessary, the precipitate is redissolved in hydrochloric acid before washing, and precipitated with ammonia). The precipitate is dried at 107 °C and calcined for one hour at 800 °C. The product should be ground to a fine powder, rewashed until free of chloride, and calcined at 1000 °C. After calcination, the TiO<sub>2</sub> so prepared is white or light gray. A yellow tinge indicates traces of iron.

Alternate methods:

II. A more readily filtered precipitate is obtained when the precipitation is carried out in the presence of  $(NH_4)_2SO_4$ . Commercial TiCl<sub>4</sub> (900 g.) is slowly added to one liter of distilled water, and the solution is purified by boiling for 10 minutes and removing SiO<sub>2</sub> and any insoluble impurities by filtration. A solution of 1300 g. of  $(NH_4)_2SO_4$  in two liters of distilled water acidified with 25 ml. of conc. hydrochloric acid is treated in a similar manner. The two solutions are cooled, combined with stirring, and brought to a boil. The pH is adjusted to 1.0 by addition of

ammonia. At pH<1, the yield is lower, while at pH>1 the Fe content of the product may exceed 0.003%. Further treatment is the same as in (I). The yield is almost quantitative, and a rutile powder with a TiO<sub>2</sub> content exceeding 99.8% is obtained.

III. 
$$K_2 \operatorname{TiF}_6 \to \operatorname{TiO}_2$$
  
240 1 79.9

A solution of  $K_2 TiF_e$  (which has been recrystallized several times) in hot water is prepared, and ammonia is added to precipitate the snow-white  $TiO_2 \cdot aq$ . The precipitate is thoroughly washed, dried and calcined.

IV. 
$$\begin{aligned} \text{TiOSO}_4 \cdot 2 \text{ H}_2\text{O} \rightarrow \text{TiO}_2 \\ 196.0 \quad 79.9 \end{aligned}$$

This may be achieved by hydrolysis of titanium sulfate solutions on prolonged boiling. However, this procedure is not recommended since it requires a long time (eight hours) and the resultant precipitates are difficult to filter; precipitation with ammonia at the boiling point is preferred.

V.  $TiCl_4 + O_2 = TiO_2 + 2Cl_2$ 189.7 32.0 79.9 143.8

The following procedure for the preparation of rutile differs fundamentally from the previous methods. Absolutely dry  $O_2$  and TiCl<sub>4</sub> vapor are passed for 20 hours through a 20-mm.-I.D. porcelain tube heated to 650-750°C. Colorless to light-yellow, lustrous crystals of rutile are deposited on the white reactor walls. Unreacted TiCl<sub>4</sub> is collected in a receiver cooled with ice-salt. Toward the end of the preparation, pure  $O_2$  is passed through the tube, and this stream is continued while the mixture is cooling.

SYNONYM:

Titanium dioxide.

PROPERTIES:

Rutile: type C4, d 4.22; anatase: type C5, d 4.06; brookite: type C21, d 4.13. M.p. 1870°C; thermal dissociation above 1800°C is evident from the appearance of a bluish tinge and a lower melting point.

Amorphous  $TiO_2$  is also insoluble in water and dilute acids. It dissolves slowly in hot conc.  $H_2SO_4$ , better in alkali hydrogen sulfates. The solubility is strongly dependent on the prior thermal treatment.

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# Titanium (IV) Oxide Hydrate

# $TiO_2 \cdot n H_2O$

I. Compounds belonging to the system  $TiO_2$ -H<sub>2</sub>O prepared in the usual way (e.g., by precipitation with ammonia from an aqueous solution of K<sub>2</sub>TiF<sub>6</sub>) may be regarded as composed of TiO<sub>2</sub> and labile H<sub>2</sub>O. Part of the water, however, is bound and localized; its amount depends on the methodof preparation. Precipitated, hydrated TiO<sub>2</sub> either exhibits an amorphous x-ray pattern or consists of anatase containing adsorbed water; similar lattices are formed by the products of hydrolysis of Ti(SO<sub>4</sub>)<sub>2</sub> solutions (refluxing for four hours), while hydrolysis of TiCl<sub>4</sub> and Ti(NO<sub>3</sub>)<sub>4</sub> solutions under identical conditions yields rutile.

II. "Orthotitanic acid"  $H_4 TiO_4$  or  $TiO_2 \cdot 2 H_2O$  seems to form only under certain definite conditions; using the Wilstätter acetone method at low temperature (0°C), it was possible to prepare a compound of composition  $TiO_2 \cdot 2.16H_2O$ .

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# Peroxotitanic Acid

# H4TiO5

According to Schwarz and coworkers, 5 g. of titanic acid hydrate (Merck) is dissolved in 10 ml. of warm conc.  $H_2SO_4$ ; the solution is diluted to three times its volume with water, cooled to  $-10^{\circ}$ C, placed in a dropping, funnel, and added to a solution of 8.6 g. of  $K_2SO_4$  in 15 ml. of 30%  $H_2O_2$ . The mixture is cooled to 0°C and allowed to stand in the cold for 1/2 hour; it is then precipitated by addition of about one liter of ice-cold acetone pretreated with  $H_2O_3$  until the appearance of the color of titanium sulfate (alcohol may cause partial reduction of the solution, yielding a product deficient in active oxygen). The precipitate is filtered with suction and washed with ice-cold absolute ether until the filtrate gives a negative reaction with permanganate. The product is dried for several hours in high vacuum at the lowest possible temperature, yielding yellow-red potassium peroxytitanyl sulfate corresponding to the formula  $K_2[TiO_2(SO_4)_2] \cdot 3 H_2O_2$ .

According to K. F. Jahr (see FIAT-Review, Anorganische Chemie, Part III, p. 173) the color is due not to the complex anion, but to the peroxytitanyl cation itself. See also E. Gastinger, Z. anorg. allg. Chem. <u>275</u>, 331 (1954).

In the preparation of the corresponding zirconium and hafnium salts, which are white but have an analogous structure, the indicated concentrations of the reactants must be very strictly adhered to.

If the complex salt is to be used immediately, purification by thorough washing suffices. The precipitate is dissolved on the filter in ice water and the solution is poured into 10 liters of icecold water. Gradual deposition of the pure white precipitate sets in after some time and the precipitation is complete after about 24 hours. The product is purified by filtering, washing with ice water followed by acetone; any adsorbed water is removed by agitating in a shaker flask (three times with 100 ml. of acetone, once with 100 ml. of absolute ether, three times with petroleum ether—all washing operations at  $0^{\circ}$ C). The remaining petroleum ether is removed by suction and the product is left for about 0.5 hour in a vacuum desiccator which does not contain a drying agent.

The peroxide hydrates of Zr, Hf and Th are prepared by treating a solution of the sulfate (5-10%, based on the oxide content) with an excess of 30% H<sub>2</sub>O<sub>2</sub>, cooling to -20°C, and precipitating with ammonia below 0°C. The slimy precipitate is removed by suction from the cooled container with the aid of a filter candle and washed with ice water. The only suitable drying method is the ammonia extraction process described by W. Blitz [Z. Elektrochem. 33, 491 (1927)].

PROPERTIES:

 $Ti(OH)_3 \cdot OOH$ : Slightly hygroscopic, lemon-yellow powder. Gradually loses its active oxygen at room temperature, with resultant decoloration. Readily soluble without decomposition in dilute H<sub>2</sub>SO<sub>4</sub>; loses oxygen gradually in water.

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# Zirconium (IV) Oxide

### ZrO<sub>2</sub>

 $\begin{array}{rrr} ZrOCl_2 \cdot 8 \ H_2O \ \rightarrow \ ZrO_2 \ \textbf{;} & Zr(SO_4)_2 \cdot 4 \ H_2O \ \rightarrow \ ZrO_2 \\ 322.3 & 123.2 & 355.4 & 123.2 \end{array}$ 

Zirconium (IV) oxide is formed when zirconium oxide hydrates or zirconium salts of volatile, oxygen-containing acids (nitrates, oxalates, acetates, etc.) are dehydrated and then calcined.

Usually the oxychloride or sulfate is thermally decomposed between 600 and 1000°C. Either salt must be prepurified by repeated recrystallization. In the case of the sulfate, the thermal decomposition removes the last traces of  $SO_3$  with some difficulty.

The amorphous  $ZrO_2$ , which is the first product obtained on heating the oxychloride (300°C), converts at 500°C to the tetragonal modification, which then contains only traces of Cl. Above 600°C the material is monoclinic.

Alternate methods: a) For the almost complete decomposition of  $\text{ZrOCl}_2 \cdot 8\text{H}_2\text{O}$  with superheated steam (accompanied by evolution of HCl and formation of  $\text{ZrO}_2$ ) see Akhrap-Simonova.

b) The preparation of  $ZrO_2$  by removal of silicon from  $ZrSiO_4$  with SiO is described by Zintl et al.

PROPERTIES:

White powder. M.p. 2680°C, b.p. 4300°C; d 5.73. Exists in several modifications. Crystal structure: tetragonal and monoclinic.

The chemical behavior is strongly affected by the nature of the prior thermal treatment. If the compound has been heated to moderate temperatures, it dissolves quite readily in mineral acids; after heating to high temperatures, it is soluble only in hydrofluoric acid and conc.  $H_2SO_4$ ; after melting, it is attacked only by hydrofluoric acid. Decomposes readily in alkali hydroxide or carbonate melts, in which it forms acid-soluble zirconates.

**REFERENCES**:

O. Ruff et al. Z. anorg. allg. Chem. <u>133</u>, 193 (1924); <u>180</u>, 19 (1929);
W. M. Cohn and S. Tolksdorf. Z. phys. Chem. (B) <u>8</u>, 331 (1930);
G. L. Clark and D. H. Reynolds. Ind. Eng. Chem. <u>29</u>, 711 (1937);
L. K. Akhrap-Simonova. Zh. Prikl. Khimii <u>11</u>, 941 (1938); E. Zintl, W. Bräuning, H. L. Grube, W. Krings and W. Morawietz. Z. anorg. allg. Chem. <u>245</u>, 1 (1940); A. W. Henderson and K. B. Higbie. J. Amer. Chem. Soc. 76, 5878 (1954).

See also R. Fricke, Das System  $ZrO_2/H_2O$  in R. Fricke and G. F. Huttig, Hydroxyde und Oxydhydrate [Hydroxides and Oxide Hydrates], Leipzig, 1937, p. 219, especially for the formation of oxide hydrates.

## Hafnium (IV) Oxide

### HfO<sub>2</sub>

Hafnium (IV) oxide is prepared by calcination of the hydroxide, oxalate, oxychloride or sulfate at 600-1000 °C. The crystallization of the oxide starts at 400 °C.

PROPERTIES:

White powder. M.p. 2780°C; d 9.68. Essentially identical with  $ZrO_2$  in chemical behavior. It probably forms the same types of crystal lattice.

**REFERENCE**:

G. von Hevesy and V. Berglund. J. Chem. Soc. (London) <u>125</u>, 2373 (1924).

### Thorium (IV) Oxide

ThO<sub>2</sub>

$$\begin{array}{ccccccc} I_{\bullet} & Th(NO_3)_4 \cdot 4 \ H_2O & {\rm or} & Th(NO_3)_4 \cdot 5 \ H_2O & {\rm or} & Th(C_2O_4)_2 \cdot 6 \ H_2O \rightarrow ThO_2 \\ & 552.2 & 570.2 & 516.3 & 264.1 \end{array}$$

Thorium (IV) oxide is obtained by thermal decomposition of thorium oxide hydrate (which is precipitated with ammonia) or salts of oxygen-containing acids. The nitrate and oxalate are especially suitable as starting materials, while the sulfates give off the last traces of  $SO_3$  only with difficulty.

Pure nitrate is placed in a large evaporation dish and is very carefully heated in an electric furnace. The nitrate swells considerably at 300-400°C and forms a spongy mass, which subsequently collapses and becomes more compact. To prevent the uptake of  $SiO_2$  during calcination of the oxide, the powder obtained on decomposition of the nitrate is placed in a Pt crucible and is then heated for 1-2 hours at 800-850°C.

The oxalate  $\text{Th}(C_2O_4)_2 \cdot 6H_2O$  gives off its water of crystallization at 300°C and is almost entirely decomposed to the oxide (the weight of the final products is <1% greater than the theoretical) at 450°C.

II. According to Brintzinger and Möllers, active oxide is obtained when thorium chloride, nitrate or sulfate is decomposed with steam at 800°C.

III. For the preparation of oxide hydrates and hydroxides, see the references indicated.

PROPERTIES:

White powder, compact or loose depending on the method of preparation. M.p.  $3050^{\circ}$ C, b.p.  $4400^{\circ}$ C; d 9.87. Crystal structure: type C1 (fluorite type). Almost insoluble in acids when calcined at high temperatures, although readily decomposed in bisulfate melts or by evaporation with conc. H<sub>2</sub>SO<sub>4</sub>. In contrast to TiO<sub>2</sub> and ZrO<sub>2</sub>, does not form salts with basic oxides and is therefore insoluble in molten NaOH or Na<sub>2</sub>CO<sub>3</sub>. The oxide prepared by calcination of the oxalate at 500°C may be dissolved by peptization with dilute hydrochloric acid.

**REFERENCES:** 

- I. J. W. Marden and H. C. Rentschler. Ind. Eng. Chem. <u>19</u>, 97 (1927); H. J. Born. Z. phys. Chem. (A) <u>179</u>, 256 (1937).
- II. H. Brintzinger and A. Möllers. Z. anorg. Chem. 254, 343 (1947).
- III. R. Fricke. Das System ThO<sub>2</sub>/H<sub>2</sub>O in: R. Fricke and G. F. Hüttig, Hydroxyde and Oxydhydrate [Hydroxides and Oxide Hydrates], Leipzig, 1937, p. 228; W. Biltz. Z. anorg. allg. Chem. <u>244</u>, 281 (1940); M. Dominé-Bergès. Ann. Chimie [12] <u>5</u>, 106 (1950).

Titanium, Zirconium and Thorium Sulfides

TiS<sub>3</sub>, TiS<sub>2</sub>, TiS<sub><2</sub>

TiS<sub>3</sub>, TiS<sub>2</sub>

I.

Titanium (IV) sulfide  $TiS_2$  is usually prepared by the reaction of a gaseous mixture of  $TiCl_4$  and  $H_2S$  in a red-hot tube. A chlorine-free product cannot be obtained without an aftertreatment with S in a pressure tube at 600°C. The yield is also unsatis-factory (30-40%, based on the TiCl<sub>4</sub>); however, the yield may be increased by repeated passage of the unreacted TiCl<sub>4</sub> from the receivers.

The reaction is carried out in the apparatus shown in Fig. 301; it consists of a Pyrex combustion tube fused at both ends to 100-ml. round-bottom flasks. Flask a is filled with 50 g, of freshly distilled TiCl<sub>4</sub> and is then sealed off at c; b is a receiver for unreacted TiCl<sub>4</sub>, and is cooled with ice-salt. A Stock receiver cooled with Dry Ice and containing about 25 ml. of liquid H<sub>2</sub>S is connected to the system at d via a small wash bottle filled with glycerol (this is the bubble counter) and two CaCl<sub>2</sub> drying tubes. A fast stream of H<sub>2</sub>S, generated by gradual removal of the coolant, is passed through the apparatus while the combustion tube is heated to  $480-540^{\circ}$ C. The TiCl<sub>4</sub> in *a* is then heated almost to the boiling point and held at this temperature with a small flame. The TiCl<sub>4</sub>-H<sub>2</sub>S gas mixture reacts in the tube to form HCl and  $TiS_2$ ; the latter settles on the tube wall. After all the  $TiCl_4$  is distilled from a, the system is flushed for a short time with  $H_2$  or  $CO_2$ , the H<sub>2</sub>S line is reconnected at e, and the unreacted TiCl<sub>2</sub> condensed in b is passed again through the tube, this time into a. After the  $TiCl_4$  has been used up (3-5 hours), the  $TiS_2$  produced is treated for two hours at the same temperature in a slow stream of H<sub>2</sub>S to remove most of the chloride still adhering to the product. The material is allowed to cool in a stream of  $H_2$  or CO<sub>2</sub>, the tube is broken at both ends, and the dark brass-yellow TiS<sub>2</sub>, which crystallizes as leaflets of mosaic gold color, is collected. The yield is about 10 g.



Fig. 301. Preparation of titanium disulfide.

Complete removel of the chloride can be effected only by repeated heating of the product with excess S in a pressure tube. Thus, 4 g. of crude  $TiS_2$  and 3 g. of S are placed in a bomb tube of Vycor or similar glass. The sulfur should be completely free of carbon compounds [von Wartenberg method: Z. anorg. allg. Chem. <u>251</u>, 166 (1943); this consists of vacuum distillation, followed by 48-hour heating at 200°C in vacuum or under N<sub>2</sub>, followed by another vacuum distillation]. The bomb

tube with the  $\text{TiS}_2$  and S is sealed in high vacuum and heated for three days at 600°C. The tube is opened and the volatile components are removed in a heated vacuum desiccator (drying pistol) at 100-150°C. The product, which still contains some chloride, is again heated with additional S to about 600°C for two days, and the volatile components are again removed in vacuum. The intermediate product then consists of graphitelike trisulfide and unreacted S. The free sulfur may be removed by vacuum distillation at 400°C; the residual TiS<sub>3</sub> undergoes thermal decomposition at temperatures above 500°C, yielding pure TiS<sub>2</sub>.

II.

 $Ti + 2S = TiS_2$ ;  $Ti + 3S = TiS_3$ 47.9 64.1 112.0 47.9 96.2 144.1

Sulfides of any desired composition may be obtained by synthesis from the elements, which is a generally applicable method. This is also the simplest way to prepare chlorine-free products. The starting material consists either of Ti filings made by grinding Ti strips (for the preparation of the strips, see the directions for the lower titanium oxides, p. 1214) or simply Ti powder (which, however, usually has a lower metal content). First 1.5 g. of Ti and 4 g. of S, in a thick-wall Vycor tube, are carefully degassed in high vacuum. The tube is then sealed and heated at 650 °C for four days. Metal particles still present in the product are ground separately, added to the sulfide product (total about 3 g.) together with 1.7 g. of S, and again heated in a pressure tube for two days at 600 °C. The free sulfur is distilled off at 400 °C; the higher sulfides are, if desired, decomposed thermally above 500 °C, yielding TiS<sub>a</sub> and TiS<sub>a</sub>, as in method I.

PROPERTIES:

 $TiS_2$ : Formula weight 112.02. Brass-yellow flakes with a metallic luster. d 3.22. Crystal structure: type C6.

Stable in air at normal temperatures; forms  $TiO_2$  on heating. Decomposes in nitric acid and hot conc.  $H_2SO_4$ , releasing S; dissolves in boiling aqueous sodium and potassium hydroxides, forming alkali titanates and alkali sulfides.

 $TiS_3$ : Formula weight 144.08. Graphitelike substance. d 3.22. Except for its insolubility in boiling NaOH, it is similar to  $TiS_3$  in all chemical properties.

 $TiS_{<2}$ 

Lower Ti sulfides may be prepared by synthesis from the elements, by treatment of Ti metal with  $TiS_2$ , or by reduction of  $TiS_2$  with  $H_2$ .

1224

### SYNTHESES STARTING FROM TITANIUM METAL

The procedure is identical to that of method II; in the first stage the S is bound to the Ti, and the resultant product is subsequently homogenized at high temperature. As long as the presence of sulfur is still a possibility, the temperature is raised very slowly, so that it may require as long as two days to reach  $800^{\circ}$ C in the case of S-rich compounds and  $1000^{\circ}$ C for S-poor compounds. The above temperatures are then maintained for 2-3 additional days, after which the product is tested for homogeneity by x-ray analysis.

The reaction is carried out in a quartz tube, which undergoes only slight devitrification but no further changes. Titanium metal itself begins to react with quartz at about  $850 \,^{\circ}$ C. To exclude entirely the possibility of reaction of the titanium with the quartz in the case of the subsulfides (TiS<sub><1</sub>), the reaction mixture is placed in small ceramic or sintered clay crucibles (8-mm. diameter, 30 mm. long) which are then inserted in suitable quartz tubes and the tubes sealed off. This complication usually results in the necessity of using tubes of lesser wall thickness; hence, greater care must be exercised during heating.

As in other cases where tubes are to be heated to temperatures as high as 800°C, protection against explosion is recommended; this is provided by wrapping the tube in asbestos paper and inserting it into a small cage made of several layers of Ni or Cu wire mesh.

When quartz tubes are cut open with an emery wheel (1 mm. thick), it is not always possible to prevent quartz splinters from getting into the product. If the material is not a mass with a solid, glossy surface affording easy visual separation, the embedded quartz particles should always be removed by shaking with bromoform followed by centrifugation.

### REDUCTION OF TiS<sub>2</sub> WITH H<sub>2</sub>

This method permits carrying the reduction as far as the monosulfide stage, provided high temperatures are used. Since all small amounts of Cl present are removed during the reduction, the crude TiS<sub>2</sub> produced by method I can be used as such without further purification. The reaction is carried out in an unglazed porcelain tube, with the boat containing the material placed in the center; a stream of carefully dried H<sub>2</sub> (freed of O<sub>2</sub> by contact with Pd) is passed over the boat. Two to three hours suffice for the reduction; the duration of the run depends on the quantity of material and the hydrogen flow rate. The temperature to which the tube is heated is a deciding factor for the degree of reduction  $(500^{\circ}C \rightarrow TiS_{1.55}, 900^{\circ}C \rightarrow TiS_{1.25}, 1200^{\circ}C \rightarrow TiS_{1.1})$ . PROPERTIES:

Air stable, colored substances ( $TiS_{1,5}$ : black;  $TiS_{1,13}$ : violet;  $TiS_{1,0}$ : 'brown;  $TiS_{0,5}$ : gray). With decreasing sulfur content, the susceptibility to hydrochloric acid attack increases and that by acidic oxidation agents decreases. In contrast to  $TiS_2$ , the lower sulfides are unaffected by sodium hydroxide. The sesquisulfide phase has a wide range of compositions ( $TiS_{1,52}$ - $TiS_{1,13}$ ).

REFERENCES:

W. Biltz and P. Ehrlich. Z. anorg. allg. Chem. <u>234</u>, 97 (1937); see also H. Hahn and B. Harder. Ibid. <u>288</u>, 241 (1956) (also contains information on growing of single crystals).

# Zirconium Sulfides

Zirconium sulfides can be prepared by exactly parallel methods, i.e., reaction of  $\text{ZrCl}_4$  with  $\text{H}_2\text{S}$  or synthesis from the elements. Orange-red  $\text{ZrS}_3$  may be thermally decomposed to brown  $\text{ZrS}_2$ at 800°C. The lower zirconium sulfides include, in addition to the sesquisulfide and subsulfide phases, an additional compound  $\text{ZrS}_{\sim 0.75}$ .

**REFERENCE:** 

E. F. Strotzer and W. Biltz. Z. anorg. allg. Chem. <u>242</u>, 249 (1939).

# Thorium Sulfides

Synthesis from the elements under pressure yields a deep red polysulfide  $Th_3S_4$ , lilac brown  $ThS_2$ , a sesquisulfide, and a subsulfide  $ThS_{0.75}$ .

REFERENCE:

E. F. Strotzer. Z. anorg. allg. Chem. 247, 415 (1941).

# Titanium (III) Sulfate

# Ti<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub>

 $\begin{array}{rrr} 2 \operatorname{TiCl}_4 \ \rightarrow \ \operatorname{Ti}_2(\mathrm{SO}_4)_3 \\ 379.5 & 384.0 \end{array}$ 

Titanium tetrachloride (100 g.) is carefully decomposed with approximately four times its volume of  $H_2O$ . The solution is cooled

and treated with dilute ammonia to precipitate  $TiO_2 \cdot aq.$ , which is suction-filtered, thoroughly washed with water, and dissolved (vigorous shaking) in 70 ml. of conc.  $H_2SO_4$ . The solution is diluted with water to one liter, the precipitation with ammonia is repeated, and the resultant deposit is reprecipitated two additional times. The  $TiO_2 \cdot aq.$  obtained (about 480 g.) is carefully dissolved in 100 ml. of conc.  $H_2SO_4$ , yielding a total liquid volume of about 500 ml.; this is suction-filtered through glass and treated with 300 ml. of conc.  $H_2SO_2$ , yielding about 400 ml. of a solution of Ti (IV) sulfate. This solution is reduced to Ti (III) sulfate by the following electrolytic method.

A low vertical cylinder closed off with a rubber stopper serves as the electrolysis vessel and contains the sulfate solution. The anode is a piece of Pt sheet immersed in a clay cell filled with 20% H<sub>2</sub>SO<sub>4</sub>. The cell is partially immersed in the Ti (IV) sulfate solution and is surrounded by four amalgamated lead strips, also immersed in the solution. The stopper on the outer electrolysis vessel has holes for the clay cell and for the inlet and outlet gas tubes. The electrolysis is carried out in a constant stream of CO<sub>2</sub> and with efficient water cooling. The current density is 0.06 amp./cm.<sup>2</sup> at 24 v. for the first six hours, then 0.33 amp./cm.<sup>2</sup> at the same voltage for an additional six hours. This reduces all the Ti (IV) sulfate to Ti (III) sulfate; the latter precipitates as an H<sub>2</sub>SO<sub>4</sub>-containing hydrate (fine, pale light violet crystals).

To obtain the anhydrous Ti (III) sulfate, the product is suctionfiltered in a stream of  $CO_2$ , washed with 50% H<sub>2</sub>SO<sub>4</sub>, and placed (in the absence of air) in a round-bottom flask fitted with a ground joint and filled with CO<sub>4</sub>. Then 300 ml. of dilute sulfuric acid  $(20\% v./v. H_2SO_4)$  is added; the flask is stoppered with a ground stopper fitted with inlet and outlet gas tubes and an opening for the insertion of a thermometer; it is heated in a stream of CO<sub>2</sub> until the precipitate dissolves. Using gas pressure, the liquid is forced into a filtration apparatus (see Part I, p. 74) and filtered under CO<sub>2</sub> through a tubular fritted glass filter fitted with appropriate ground joints. The receiver with the filtrate is in turn closed off with a stopper fitted with a thermometer and gas tubes; CO<sub>2</sub> is passed through, and the temperature is slowly raised by means of an oil bath. This concentrates the liquid to about half its volume. At this point a violet precipitate begins to form; on further heating, this turns to blue and finally, at 190-200°C, to green. The temperature is maintained at 190°C for three hours and is then raised to 210-220°C for 10 minutes. Heating to higher temperatures results in evolution of SO<sub>2</sub> and oxidation of the Ti (III) sulfate. The material is allowed to cool in a stream of CO<sub>2</sub>; it should remain green. If it assumes a blue color, the heating procedure must be repeated. The green precipitate is filtered under CO<sub>2</sub>, using the filtration apparatus; the contaminating Ti (IV) sulfate

is removed by washing with conc.  $H_2SO_4$ , followed by glacial acetic acid, anhydrous methanol and ether. The tubular fritted glass filter is removed from the filtration system, covered with a ground cap fitted with a stopcock, and placed horizontally in a short tubular electric furnace; the material is then dried for four hours at 140°C in a constant stream of pure N<sub>2</sub>.

PROPERTIES:

Green crystalline powder. Insoluble in water, alcohol and conc.  $H_2SO_4$ ; soluble in dilute  $H_2SO_4$  and in hydrochloric acid, yielding a violet solution.

REFERENCES:

 O. Schmitz-Dumont, P. Simons and G. Broja. Z. anorg. Chem. <u>258</u>, 307 (1949); W. J. de Haas and B. H. Schultz. Physica [2] <u>6</u>, 481 (1939); A. Stähler and H. Wirthwein. Ber. dtsch. chem. <u>Ges. 38</u>, 2619 (1905).

# Titanoxy Sulfate

### TiOSO4

This compound is produced on evaporation of  $\text{TiO}_2$  or  $\text{TiO}_2 \cdot \text{aq.}$ with conc.  $\text{H}_2\text{SO}_4$ ; the dihydrate is obtained under the same conditions but with 70%  $\text{H}_2\text{SO}_4$ . The material, which is extremely hygroscopic and readily splits off SO<sub>3</sub>, can also be prepared as a white precipitate by dropwise addition of a solution of SO<sub>3</sub> in SO<sub>2</sub>Cl<sub>2</sub> to a solution of TiCl<sub>4</sub> in SO<sub>2</sub>Cl<sub>2</sub>, followed by refluxing [E. Hayek and W. Engelbrecht, Monatsh. Chemie <u>80</u>, 640 (1949)].

Iron-free titanoxy sulfate is usually not available commercially. When available, it is not completely water soluble. The following procedures are therefore recommended for the preparation of the pure compound.

I. TiOSO4

$$TiCl_4 + H_2SO_4 + H_2O = TiOSO_4 + 4 HCl$$
  
189.7 98.1 18.0 160.0 145.8

Pure, multiple-distilled TiCl<sub>4</sub> is added in drops (vigorous stirring) to the stoichiometric quantity of 50% H<sub>2</sub>SO<sub>4</sub>. The precipitate formed after each addition should dissolve completely before the next portion of TiCl<sub>4</sub> is added. After the addition of about 3/4 of the TiCl<sub>4</sub>, the liquid turns into a viscous, yellowish solution. It

is diluted with 1/5 its volume of water, and the dropwise addition of TiCl<sub>4</sub> is completed.

The resultant solution, which is still highly concentrated in HCl, is evaporated to dryness on a water bath. The residue is pulverized, dried and freed of HCl by heating for several days at 80-100°C in a drying pistol at a pressure of a few mm.

The sulfate obtained in this manner  $(TiO_2:SO_3 = 1:1.07)$  is colorless and free of HCl. It is hygroscopic and soluble in water, yielding a clear solution.

Gelatinous or resinous precipitates may appear during the evaporation of the HCl-containing, highly viscous and slightly yellow solutions; the same phenomenon may occur during the vacuum concentration operation. Addition of alcohol or acetone to the concentrated solutions leads to the formation of fibrous precipitates.

II.  $TiOSO_4 \cdot 2 H_2O$ 

A solution of 40 ml. (63 g.) of freshly distilled TiCl<sub>4</sub> (b.p. 134-138°C) in 130 ml. of water is prepared. Most of this solution is then reduced to a slight extent by means of Zn rods or, better, by electrolysis (light-brown color), while 10 ml. kept separately is reduced to the trivalent titanium ion (deep violet color). The reduction is intended to ensure that all the iron is present in the form of  $Fe^{2+}$ , to avoid hydrolysis and coprecipitation of  $Fe^{3+}$  with the Ti. The reaction is carried out in dilute oxalic acid.

The violet chloride solution is slowly added in drops to a boiling solution of 1 g. of oxalic acid in one liter of water; then the brown chloride solution is added in the same way. The mixture should be maintained at the boiling point for a total of four hours, the volume being kept constant by occasional addition of water. The conditions of precipitation must be closely adhered to, to prevent coprecipitation of unfilterable metatitanic acid. The precipitate is filtered through a large Büchner funnel, washed free of Cl with boiling water and dried by suction. It is then treated with 35 ml. of conc.  $H_2SO_4$  in a beaker. The mixture is gently boiled until precipitation begins. After cooling and addition of 120 ml. of water, the mixture is allowed to stand (with occasional stirring) for several days and, if necessary, is filtered. It is then evaporated, precipitating crystals of TiOSO<sub>4</sub> · H<sub>2</sub>O. III. TiOSO<sub>4</sub> solutions

 $\begin{array}{rll} TiO_2 &+& 2\,KHC_2O_4 &+& H_2O &=& K_2TiO(C_2O_4)_2\cdot 2\,H_2O \\ &&& 79.9 & 256.3 & 18.0 & 354.2 \\ K_2TiO(C_2O_4)_2\cdot 2\,H_2O &+& 2\,H_2SO_4 &=& \\ && 354.2 & 196.2 \\ && K_2SO_4 &+& TiOSO_4 &+& 2\,CO &+& 2\,CO_2 &+& 4\,H_2O \\ && 174.3 & 160.0 & 56.0 & 88.0 & 72.1 \end{array}$ 

Iron-free titanium sulfate solutions, used as analytical standards, are readily prepared by repeated recrystallization of  $K_2TiO(C_2O_4)_2 \cdot 2H_2O$  followed by treatment with conc.  $H_2SO_4$ .

A concentrated solution of  $KHC_2O_4$  is saturated at the boiling point with freshly precipitated  $TiO_2 \cdot aq.$ ; the mixture is concentrated, whereupon white needles precipitate out. The double salt is dissolved with heating in an approximately equal weight of water. The solution is filtered and the salt is recrystallized in about 80-90% yield by cooling with ice and stirring. The iron content is reduced in the process from 0.061% to 0.004%, and no further iron can be detected after another repetition of the crystallization. The analytically pure salt has a composition corresponding to  $K_2TiO(C_2O_4)_2 \cdot 2H_2O$ .

To prepare one liter of an approximately 0.1N Ti sulfate solution, 38 g. of the double oxalate is thoroughly mixed with 32 g. of pure  $(NH_4)_3SO_4$  (iron-free!) and placed in a 750-ml. Kjeldahl flask. The addition of  $(NH_4)_3SO_4$  facilitates the reaction. Then 80 ml. of pure conc.  $H_3SO_4$  is added. The flask is heated carefully with a small flame until cessation of foaming, and the solution is then boiled on a strong flame to decompose the oxalate. The solution is cooled, whereupon it becomes sirupy; it is carefully added, with vigorous stirring, to 500 ml. of distilled water and is then diluted to one liter. If precipitation occurs after standing overnight, the solution is filtered. The solution should give a negative test for oxalate upon addition of 1 drop of 0.1N KMnO<sub>4</sub> to 50 ml. of the liquid.

IV. Alternate method: Pure  $TiOSO_4 \cdot 2H_2O$  precipitates in the form of long crystal needles from a solution of  $TiO_2 \cdot aq$ . in 60-70%  $H_2SO_4$  in which the ratio  $TiO_2: H_2SO_4 = 1:3$  to 1:7. At higher concentrations of acid, the precipitate is powdery and contains less  $H_2O_4$ .

SYNONYM:

Titanyl sulfate.

PROPERTIES:

Anhydrous: highly hygroscopic; dissolves slowly in water to give a clear solution. Dihydrate: readily soluble. Decomposes in hot  $H_2O$  with precipitation of  $TiO_2 \cdot aq$ .

REFERENCES:

- I. Private communication from K. F. Jahr, Berlin.
- II. A. W. Hixson and W. W. Plechner. Ind. Eng. Chem. <u>25</u>, 262 (1933).
- III. R. Rosemann and W. M. Thornton. J. Amer. Chem. Soc. <u>57</u>, 328 (1935); see also C. Péchard. Comptes Rendus Hebd. Séances Acad. Sci. <u>116</u>, 1513 (1893); A. Rosenheim and O. Schütte. Z. anorg. Chem. <u>26</u>, 239 (1901).
- IV. A. V. Pamfilov and T. A. Chudyakova. Zh. Obschey Khimii <u>19</u>, 1443 (1949).

## Zirconium Sulfates

 $Zr(SO_4)_2 \cdot 4 H_2O$ 

Hauser and more recently Falinski have made a thorough study of the system  $ZrO_2/SO_3/H_2O$  as a function of the  $SO_3$  concentration. It was found that the tetrahydrate is formed on addition of  $ZrOCl_2 \cdot 8H_2O$  to a sulfuric acid containing less than 64%  $SO_3$  (d 1.714). The tetrahydrate solubility is then 2%. The minimum solubility (0.3%) corresponds to 50%  $SO_3$  in the acid (d 1.517).

If the SO<sub>3</sub> content exceeds 64%, acid sulfates precipitate:  $Zr(SO_4)_2 \cdot H_2SO_4 \cdot 2H_2O$  at an SO<sub>3</sub> concentration of 64-72%, and  $Zr(SO_4)_2 \cdot H_2SO_4 \cdot H_2O$  at 72-79% SO<sub>3</sub>.

PROPERTIES:

Formula weight 355.41. Orthorhombic crystals. The basic salt precipitates slowly from neutral, saturated solution, rapidly from dilute solution or above 40°C.

In sulfuric acid solution, the Zr migrates to the anode instead of to the cathode; therefore, the tetrahydrate may actually be present in the form of a disulfatooxozirconic acid  $(H_2[OZr(SO_4)_2] \cdot 3H_2O)$ .

 $Zr(SO_4)_2$ 

The anhydrous salt is obtained by evaporation of the tetrahydrate or Zr oxychloride with conc.  $H_2SO_4$ . Thus, 100 g. of  $ZrOCl_2 \cdot 8H_2O$  is mixed with 50 g. of conc. H<sub>2</sub>SO<sub>4</sub> and, after termination of the gas evolution, carefully evaporated with stirring on a sand bath. The dihydrate of the sulfate crystallizes out as an intermediate in the process, but decomposes on further heating, giving off additional H<sub>2</sub>O. The excess H<sub>2</sub>SO<sub>4</sub> is completely evaporated by heating to 350-380 °C. The product is allowed to cool in a desiccator.

PROPERTIES:

Formula weight 283.35. Microcrystalline powder. Very hygroscopic; in air, forms an unstable solution from which the normal tetrahydrate crystallizes after some time. The anhydride dissolves more rapidly in a small than in a large amount of water, since the temperature rise produced by the heat of hydration sharply accelerates the solution process.

REFERENCES:

O. Hauser. Z. anorg. allg. Chem. <u>106</u>, 1 (1919); M. Falinski. Ann. Chimie <u>16</u>, 237 (1941).

# Purification of Zr salts via the tetrahydrate

One volume of conc.  $H_2SO_4$  is added to two volumes of a moderately conc. aqueous solution of Zr sulfate or chloride; the thick, white, crystalline precipitate of  $Zr(SO_4)_2 \cdot 4H_2O$  is readily filtered with suction on fritted glass of medium porosity. Since 1 g. of the salt dissolves in 1 ml. of  $H_2O$ , the Zr sulfate is easily dissolved; it is reprecipitated by addition of conc.  $H_2SO_4$ . The iron is efficiently removed during the recrystallization provided the solution contains about 10% HCl. After each precipitation, the solid is washed several times with a solution made up of 15 parts by volume of  $H_2O$ , eight parts of conc.  $H_2SO_4$  and one part of conc. hydrochloric acid, followed by three washings with acetone. Alcohol should not be used for washing, since it forms complexes during the further precipitations.

Thus, 1135 g. of commercial  $\text{ZrCl}_4$  [equivalent to 1731 g. of  $\text{Zr}(\text{SO}_4)_2 \cdot 4\text{H}_2\text{O}$ ] was dissolved in 1800 ml. of  $\text{H}_2\text{O}$  and 250 ml. of conc. HCl; then 100 ml. of conc.  $\text{H}_2\text{SO}_4$  was added, precipitating 1640 g. (94%) of  $\text{Zr}(\text{SO}_4)_2 \cdot 4\text{H}_2\text{O}$ ; five additional crystallizations of this product (under identical conditions) finally give 1212 g. of pure tetrahydrate. The following table shows the degree of purification achieved:

	~0.1%	··· 0.1%	~ 0.01%	~ 0,001%	< 0,0001%	-: 0,00001%
Starting material	Fe	Mg, Si	Ag, Al, Ba, Ti	Ca, Cu, Mii	_ /*	
6 x recrystallized	. —	_	_	Ca, Mg, Na, Si	$\mathbf{A}\mathbf{g}$	Fe, Cu

A residual Fe content up to 0.01% may be removed by mere recrystallization, without the addition of HCl. The Hf content of 2.7% remains unchanged.

REFERENCE:

W. S. Clabaugh and R. Gilchrist. J. Amer. Chem. Soc. <u>74</u>, 2104 (1952).

### Titanium, Zirconium and Hafnium Nitrides

#### TiN, ZrN, HfN

#### Titanium Nitride

A number of procedures are available. The simplest is the industrial process starting from  $TiO_2 + C + N_2$  described in method I. If metallic Ti is available, synthesis from the elements (method II) is recommended. Very pure nitride in rod or wire form, especially well suited for physical measurements, is obtained by vapor deposition (method III). An additional method of lesser importance consists of the reaction between  $TiCl_4$  and  $NH_3$  (method IV).

I.  $2 \operatorname{TiO}_2 + 4 \operatorname{C} + \operatorname{N}_2 = 2 \operatorname{TiN} + 4 \operatorname{CO}_{159.8} 48.0 28.0 123.8 112.0$ 

The industrial process does not yield a pure product. Acetylenederived carbon black is degassed at 1200 °C, mixed with TiO<sub>2</sub> (mole ratio 1:2), and the mixture placed in a silicon carbide or molybdenum boat; it is calcined for three hours at 1250 °C in a stream of N<sub>2</sub>. The product contains 98% nitride; the remainder is lower Ti oxides.

Ш.	2 Ti	+	$N_2$	=	$2 \operatorname{TiN}$
	95.8		28.0		123.8

In the first stage of this preparation, the ductile metal absorbs small quantities of  $N_2$  and thus becomes brittle; it can then be pulverized for further treatment with nitrogen.

#### TREATMENT WITH NITROGEN

Titanium filings ground to pinhead size and treated as described in the directions for the preparation of TiO (p. 1214) are placed in a silicon carbide or molybdenum boat and heated to 1200°C for 2-3 hours. The operation is carried out in a hard porcelain tube in a stream of pure, dry N<sub>2</sub>. The product has the approximate composition  $TiN_{0.95}$ ; it is finely ground and subjected to a second nitrogenation. The desired composition is attained in two to three repetitions of the procedure. The quality of the product is adversely and decisively affected by the presence of the slightest traces of O<sub>2</sub> or H<sub>2</sub>O during heating.

### PRESINTERING

The products prepared in the above manner are ground as finely as possible and compressed under 2000 kg./cm<sup>2</sup> into rods  $3 \times 40$ or  $5 \times 40$  mm. Successful molding usually requires the addition of 2-5% of metal powder. The rods are embedded in nitride powder (to prevent formation of an oxide coating) and presintered in a small tubular tungsten furnace (cf. Part I, p. 40) at about 2300°C in a stream of N<sub>2</sub>; the small amount of free metal is converted to nitride in the process. Since the reaction is usually accompanied by considerable shrinkage of the rods and concomitant appearance of porosity, the material must be repulverized, remolded and resintered. This procedure is repeated two to four times, until the presintered rods exhibit some constancy of density.

### HIGH-TEMPERATURE SINTERING

When the rods have attained sufficient strength and density in the presintering process, they are fastened with clamps in preparation for direct electrical heating. The operation is carried out in technical-grade Ar containing 12-15% N<sub>2</sub>; the equipment used has been described by C. Agte and H. Alterthum, Z. techn. Phys. 11, 182 (1930).

The nitrides are heated to just below their melting points. At these extreme temperatures, all impurities (except some oxides and the carbides) possess higher vapor pressures than the nitrides and therefore evaporate. However, the oxides, even though their melting points are lower, are difficult to remove. The carbides, with higher melting points than the nitrides, remain unchanged.

III. 
$$2 \operatorname{TiCl}_4 + N_2 + 4 H_2 = 2 \operatorname{TiN} + 8 \operatorname{HCl}_{379.4} 280 8.1 123.8 291.7$$

The technique used in the vapor deposition process is the same as that described for the preparation of the metals (Ti, Zr and Hf) from the gas phase, except that the gas used here is a  $TiCl_4$ saturated mixture of  $H_2 + N_2$  (the reaction at the glowing wire is less successful with  $N_2$  alone).

A gasometer is filled with a mixture of equal volumes of  $H_2$ and  $N_2$ . The gas bubbles through a 36 °C wash bottle filled with  $TiCl_4$  (p of  $TiCl_4 = 17 \text{ mm.}$ ). The gas flow rate is of no importance, except that it must be low enough to achieve saturation. The optimum reaction pressure is about 30-40 mm., measured with a manometer whose mercury surface is protected by a thin film of butyl phthalate.

The reactor is a round-bottom Pyrex flask with inlet and outlet tubes for the gas fused on the sides. The arc-shaped glow wire (about 8-10 cm. long) is welded to two thick tungsten electrodes; these are sealed into a ground joint inserted through the bottom of the flask. The 0.2-mm. glow wire may be either W or Ta. Wires of Ta can be directly welded to the W rods, whereas the W wires have to be connected via a short Ni bridgepiece.

In the course of the reaction, the glow wire is heated to about  $1450^{\circ}$ C; since the depositied TiN is itself a good electrical conductor, the current must be raised from 10 to about 22 amp. within the first 40 minutes. One serious disadvantage of the process is the fact that the temperature cannot be measured with an optical pyrometer because TiCl<sub>3</sub>, one of the products of the fast decomposition of TiCl<sub>4</sub>, soon coats the flask walls. One must resort, therefore, to indirect estimation of the temperature by measuring it in a blank run with gases containing no TiCl<sub>4</sub>. The nitride deposits as a fine crystalline coating of copper to gold luster.

Alternate methods:

IV.

 $TiCl_4 + NH_9 \rightarrow TiN$ 189.7 17.0 61.9

As we have indicated before, this process is less desirable. Chlorine-containing TiN is formed (in poor yield) on the walls of a porcelain tube in which a gaseous mixture of  $TiCl_4$  and  $NH_3$  is thermally decomposed at 800°C. The same result is obtained when the solid compound  $TiCl_4 \cdot 4NH_3$  is placed at the front end of the tube, evaporated in a stream of  $NH_3$ , and allowed to react at 800°C.

The TiCl<sub>4</sub> ammoniate is prepared by distilling the TiCl<sub>4</sub> into a bomb tube and covering the liquid with excess  $NH_3$  at -60°C; the pale yellow compound is formed after shaking the sealed tube for 12 hours at room temperature.

The "crude nitride" formed at 800°C is heated in a hard porcelain tube at 1500°C for six hours in a stream of  $NH_3$  to obtain a Cl-free product of the composition TiN.

### PROPERTIES:

Bronze-colored powder. M.p. 2950°C; d 5.21. Somewhat dissociated at the melting point. Crystal structure: type B1 (NaCl type). This structure holds for a wide range of compositions ( $TiN_{1.0}$ - $TiN_{0.4}$ ). Very good electrical conductor. Insoluble in HCl,  $HNO_3$  and  $H_2SO_4$ , even on boiling; dissolves rapidly in hot aqua regia. Evolves  $NH_3$  on boiling in potassium hydroxide and on heating in soda lime.

# Zirconium Nitride, Hafnium Nitride

These compounds are prepared by the same methods as above.

I. The experimental arrangement is the same as for TiN, except that the reaction temperature is higher (about 1300 °C). Since Mo begins to form a carbide at this temperature, the equipment must be made of tungsten. The products are only about 90% pure; the remainder is mainly the oxide.

II. Yellowish-brown ZrN (m.p. 2980 °C) is synthesized from the elements by heating the latter for two hours at 1200 °C. The corresponding temperature for HfN is 1400-1500 °C.

III. The vapor deposition method: If  $H_2$  (or  $H_2 + N_2$ , or  $NH_3$ ) is the carrier gas for  $ZrCl_4$  (or  $HfCl_4$ ) the required wire temperatures are 2000-2400°C. With pure  $N_2$ , the temperature must be 2900°C and the rate of deposition is considerably slower.

Nitridation of Zr wire by heating in pure  $N_2$  produces ZrN at very low rates and in a very loose and brittle form, even if the temperature is almost at the melting point of the metal (1860°C).

REFERENCES:

I.

- General: C. Agte and K. Moers. Z. anorg. allg. Chem. <u>198</u>, 233 (1931).
  - I. E. Friederich and L. Sittig. Z. anorg. allg. Chem. 143, 293 (1925).
- II. P. Ehrlich. Z. anorg. Chem. <u>259</u>, 1 (1949); G. L. Humphrey. J. Amer. Chem. Soc. <u>75</u>, 2806 (1953).
- III. A. E. van Arkel and J. H. deBoer. Z. anorg. allg. Chem. <u>148</u>, 345 (1925); F. H. Pollard and P. Woodward. J. Chem. Soc. (London) <u>1948</u>, 1709; Trans. Faraday Soc. <u>46</u>, 190 (1950); F. H. Pollard and G. W. A. Fowles. J. Chem. Soc. (London) <u>1952</u>, 2444.
- IV. A. Brager. Acta Physiochim. URSS <u>11</u>, 617 (1939); see also
   O. Ruff and F. Eisner. Ber. dtsch. chem. Ges. <u>41</u>, 2250 (1908); <u>42</u>, 900 (1909).

# Thorium Nitride

Th<sub>3</sub>N<sub>4</sub>

Thorium nitride is usually prepared by heating Th filings in a stream of dry, pure  $N_2$ . The reaction is complete in three

hours at 800°C. The presence of oxide in the metal is detrimental (products lower in N are formed).

#### II. Alternate methods:

a) 
$$3 \operatorname{ThCl}_4 + 2 \operatorname{N}_2 + 6 \operatorname{H}_2 = \operatorname{Th}_3 \operatorname{N}_4 + 12 \operatorname{HCl}_{112.2} 5.6 1.2 75.2 43.8$$

Since  $Th_3N_4$  is not an electrical conductor, the method of vapor deposition (used in the preparation of pure TiN and ZrN) gives poorer results. Thus, solid  $ThCl_4$  is made to react with  $N_2 + H_2$  in a flask maintained at 800°C. The tungsten glow wire is at < 1000°C. The yield is poor.

b) 
$$3 \operatorname{ThO}_2 + 6 \operatorname{C} + 2 \operatorname{N}_2 = \operatorname{Th}_3 \operatorname{N}_4 + 6 \operatorname{CO}_{792.4} 72.1 56.0 752.4 168.1$$

A sintered tungsten rod containing  $ThO_2$  + graphite is calcined above 2000°C in a N<sub>2</sub>-containing atmosphere; black  $Th_3N_4$  crystals and lighter, oxide-containing products are formed.

#### PROPERTIES:

Dark-brown, almost black powder. Stable in dry air; readily soluble in acids.

REFERENCES:

- I. B. Neumann, C. Kröger and H. Haebler. Z. anorg. allg. Chem. 207, 145 (1932).
- II. W. Düsing and M. Hüniger. Techn. Wissenschaftl. Abhdlg. Osram 2, 357 (1931).

#### **Titanium Tetranitrate**

#### Ti(NO<sub>3</sub>)<sub>4</sub>

I,

$$\begin{array}{rrrr} \text{TiCl}_4 &+ & 4 \text{ N}_2 \text{O}_5 &= & \text{Ti}(\text{NO}_3)_4 &+ & 4 \text{ NO}_2 \text{Cl} \\ 189.7 & & 432.0 & & 296.0 & & 325.7 \end{array}$$

A solution of 3 ml. (5.1 g.) of  $TiCl_4$  in 10 ml. of  $CCl_4$  is cooled with Dry Ice in a two-neck flask provided with a dropping funnel and a reflux condenser which is protected with a  $P_2O_5$  tube. A solution of 11.6 g. of  $N_2O_5$  in 25 ml. of  $CCl_4$  is then added dropwise. A yellow, flocculent precipitate forms as soon as the  $CCl_4$  melts (-23°C), that is, after removal of the coolant. As the temperature rises from -23 °C to room temperature, the precipitate dissolves with evolution of a gas. If too violent, the bubbling may be slowed down by cooling. Solution of the last fraction of the precipitate is accelerated by mild heating. The volatile components (NO<sub>2</sub>Cl and CCl<sub>4</sub>) are removed by vacuum distillation, leaving a residue of the white Ti(NO<sub>3</sub>)<sub>4</sub>, which may be purified by sublimation in high vacuum at 50 °C. Part of the product decomposes in the process into N<sub>2</sub>O<sub>5</sub> and nonvolatile TiO(NO<sub>3</sub>)<sub>2</sub>.

II.

 $TiCl_4 + 4 ClNO_3 = Ti(NO_3)_4 + 4 Cl_2$ 189.7 389.9 296.0 283.6

A large excess of  $CINO_3$  is condensed at liquid nitrogen temperature onto the surface of 2-3 ml. of  $TiCl_4$ , frozen in a cold trap at high vacuum. The trap is connected to a surge vessel and the temperature is raised to -80 °C. The reaction, which is accompanied by evolution of chlorine, is complete after a few hours. The volatile components ( $Cl_2$  and excess  $CINO_3$ ) are then distilled off in vacuum at room temperature. The  $Ti(NO_3)_4$  residue, in the form of a crystal cake, may be further purified by sublimation at 50 °C.

PROPERTIES:

After sublimation slightly above the melting point, white needles. M.p. 58.5°C. Decomposes at about 100°C.

**REFERENCES:** 

- I. M. Schmeisser. Angew. Chem. <u>67</u>, 493 (1955); D. Lützow, Thesis, Univ. of Munich, 1955.
- II. W. Fink. Thesis, Univ. of Munich, 1956.

### Thorium Nitrate

#### Th(NO<sub>3</sub>)<sub>4</sub> · nH<sub>2</sub>O

#### RECOVERY OF THORIUM SALTS FROM MONAZITE

The mechanical ore-dressing process yields a monazite sand concentrate consisting of a mixture of Th silicate, rare earth phosphates,  $SiO_2$  and usually 4-5% ThO<sub>2</sub>. The material is calcined, finely ground and dissolved by prolonged digestion with conc.  $H_2SO_4$  at 210°C. After cooling, the pasty mass is dissolved in ice water and the undissolved material is filtered off. Further treatment may proceed via the following methods.

1238

I. In this method, Th is quantitatively precipitated as the phosphate, together with a small amount of rare earth phosphates; this is accomplished by neutralization and dilution of the solution. The phosphates are dissolved in conc. hydrochloric acid and precipitated with oxalic acid, and the thoroughly washed precipitate is extracted with warm aqueous  $Na_2CO_3$ . Most of the rare earths stay in the residue, while the thorium dissolves in the form of a carbonate complex,  $Na_6Th(CO_3)_5$ . The material is freed of the remaining traces of rare earths by repeated crystallization in the form of the sulfate  $Th(SO_4)_2 \cdot 8H_2O$ . The procedure consists of precipitation of the hydroxide with ammonia and solution of the latter in sulfuric acid to re-form the sulfate. The precipitate from the last purification stage is dissolved in nitric acid to yield the nitrate.

[In an older process, the Th and the rare earths are coprecipitated as oxalates from the initial acidic solution. The Th is extracted as the carbonatothorate by treatment with aqueous  $Na_2CO_3$  solution.]

II. A good yield of ThO<sub>2</sub> may be obtained in another process used primarily for production of the rare earths. The filtered sulfate extract is treated with  $Na_2SO_4$  to precipitate the cerium earths (as double sulfates); the corresponding Th salt  $Na_2SO_4$ . Th(SO<sub>4</sub>)<sub>2</sub>. 6H<sub>2</sub>O remains in solution. The mixture is filtered and the filtrate is heated to 90°C; it is treated with oxalic acid, yielding a precipitate consisting chiefly of Th oxalate.

Further treatment is the same as in method I.

# **III. PURIFICATION OF THORIUM NITRATE**

Very pure  $NH_4NO_3$  is added to a solution of the crude nitrate; the result is the double nitrate  $Th(NO_3)_4 \cdot NH_4NO_3 \cdot 8H_2O$ . The product is further purified by dissolving in triple-distilled water, adding redistilled nitric acid, and concentrating in a Pt dish on an electrically heated water bath until crystallization begins. The solution is then cooled with ice and constantly stirred; the resultant crystals are centrifuged off and redissolved. The procedure is repeated five times, yielding about 50% of the initial Th as the double nitrate.

The product is dissolved in very pure water, filtered and precipitated as the oxalate by addition of a nitric acid solution of purified oxalic acid; the precipitate is suction-filtered, washed with alcohol, and dried. The resultant Th oxalate may be calcined immediately to the oxide, or it may be reconverted to the nitrate by dissolving in conc. nitric acid.

## IV. PREPARATION OF THE HYDRATES

Depending on the conditions of preparation,  $Th(NO_3)_4$  crystallizes from solutions of thorium hydroxide (or from HNO<sub>3</sub> solutions of moderately calcined oxide) with varying contents of water of crystallization. When a not too acid solution is concentrated by evaporation,  $Th(NO_3)_4$  crystallizes in the cold with 12 moles of H<sub>2</sub>O. A solution evaporated at 15°C yields the pentahydrate, which is stable to 80°C if heated in an atmosphere free of CO<sub>2</sub>. At higher temperatures, it converts to the trihydrate, and between 125 and 150°C, to the hemihydrate. Above 150°C the remaining water is split off, together with nitrogen oxides.

### V. PREPARATION OF THE ANHYDRIDE

Anhydrous Th(NO<sub>3</sub>)<sub>4</sub> is prepared by treatment of the lower hydrates of Th with  $N_2O_5$  condensed at  $-78^{\circ}C$ .

PROPERTIES:

Formula weights:  $Th(NO_3)_4$  480.15;  $Th(NO_3)_4 \cdot 5H_2O$  570.23. Very readily soluble in  $H_2O$  and alcohol. Due to hydrolysis, the aqueous solution becomes acid and slowly precipitates a basic salt. The commercial product usually contains about four moles of  $H_2O$ ; it usually also contains some sulfate. Combines very readily with alkali and alkaline earth nitrates to yield double nitrates (very beautiful crystals). The alkali salts corresponding to the formula  $Alk_2[Th(NO_3)_6]$  crystallize in anhydrous form, and the corresponding alkaline earth compounds with eight moles of  $H_2O$ . Watercontaining alkali thorates  $Alk[Th(NO_3)_6]$  have also been described.

**REFERENCES:** 

- I. D. W. Pearce, R. A. Hansen and J. C. Butler in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 38; see also H. and W. Biltz. Ubungsbeispiele aus der unorganischen Experimentalchemis [Laboratory Problems in Experimental Inorganic Chemistry], Leipzig, 1920, p. 226; and L. Vanino. Handbuch der Präparativen Chemie [Handbook of Preparative Chemistry], Vol. I (Inorg. Part), Stuttgart, 1925, p. 759.
- II. E. S. Pilkington and A. W. Wylie. J. Soc. Chem. Ind. <u>66</u>, 387 (1947); this article also lists additional references on the subject.
- III. O. Hönigschmid and S. Horovitz. Sitz.-Ber. Akad. Wissensch. Wien IIa, 12b, No. 3 (1916); see also E. H. Archibald. The Preparation of Pure Inorganic Substances, New York, 1932, p. 193.

- IV. E. Chauvenet and Souteyrand-Franck. Bull. Soc. Chim. France [4] <u>47</u>, 1128 (1930).
- V. P. Miscialeti. Gazz. Chim. Ital. <u>60</u>, 882 (1930); see also J. R. Ferraro, L. J. Katzin and G. Gibson. J. Amer. Chem. Soc. <u>77</u>, 327 (1955).

# Titanium Oxonitrate, Zirconium Oxonitrate

# TiO(NO<sub>3</sub>)<sub>2</sub>, ZrO(NO<sub>3</sub>)<sub>2</sub>

A suspension of  $\text{TiI}_4$  or  $\text{ZrI}_4$  in anhydrous  $\text{CCl}_4$  is placed in a three-neck flask and agitated with a magnetic stirrer. Dry dinitrogen tetroxide is then bubbled through; the excess gas and the NO formed in the reaction are allowed to escape through a  $P_2O_5$  tube. On contact with the gas, the liquid assumes a deep violet color due to liberation of iodine. After about one hour, the product is suction-filtered through a sintered glass plate; this operation is carried out in a dry box in flowing nitrogen (see Part 1, p. 71). The product is then washed with anhydrous  $\text{CCl}_4$ and the solvent removed in vacuum.

PROPERTIES:

The almost white, powdery oxonitrates are hygroscopic; on heating, they are converted to the dioxides without melting. Soluble in alcohol, insoluble in benzene and  $CCl_4$ .

REFERENCE:

V. Gutmann and H. Tannenberger. Monatsh. Chem. 87, 424 (1956).

Titanium Phosphide, Zirconium Phosphides, Thorium Phosphide

TiP, ZrP<sub>2</sub>, ZrP, Th<sub>3</sub>P<sub>4</sub>

# Titanium Phosphide

I. The process recommended for the preparation of titanium phosphides is the pressure synthesis from the elements in the "Faraday" apparatus (see Part I, p. 76).

$$Ti + P = TiP$$
  
47.9 31.0 78.9

Biltz et al. give the following procedure: 2 g. of Ti filings ground to pinhead size (for their preparation see the directions given for the lower titanium oxides, p. 1214) and 4 g. of red P are weighed into a small ceramic or sintered clay cylindrical crucible. The materials are degassed by fanning with a flame in high vacuum and are then sealed (in vacuum) into a quartz pressure tube. The colder half of the tube is heated to 450°C, while the Ti side is maintained at 950°C. Two three-day periods are needed for the reaction. After the first period, the tube is slowly cooled for 3-4 hours, and the unreacted P is thus distilled into the cooler section. The product is readily ground in an agate mortar, an indication that the ductile Ti metal has reacted. The grinding is carried out under CS<sub>2</sub>, which is then removed with alcohol and by drying in vacuum over NaOH at 120-140°C. Microscopic examination of the dark-gray metallic product should show no red phosphorus. The phosphorus quantity used for the second reaction stage should again correspond to an atomic ratio of 3 P:1 Ti. The product treatment after the second three-day heating period is the same as that after the first. The resultant phosphide does not correspond completely to the formula TiP (maximum composition is  $TiP_{0.92}$ ).

п.

$$\begin{array}{l} {\rm TiCl}_4 + {\rm PH}_3 \rightarrow {\rm TiP} \\ {\rm 189.7} \quad {\rm 34.0} \quad {\rm 78.9} \end{array}$$

The method of Gewecke starts from phosphine generated from yellow P and KOH; the gas is washed with conc. hydrochloric acid to remove spontaneously igniting phosphorus hydrides, followed by NaOH. It is dried in two U tubes filled with  $CaCl_2$  pieces (broken up from a solidified melt of the salt) and two  $P_2O_5$  tubes. The gas enters the reaction apparatus proper through a trap (-250°C) designed in such a way that any gaseous TiCl<sub>4</sub> backing up during the reaction will condense out. The PH<sub>3</sub> train is connected to the reactor via a ground joint, with a two-way stopcock (for venting) inserted in the line.

The reactor consists of a a spherical  $TiCl_4$  vessel followed by a heating tube 40 cm. long. Both the vessel and the tube are Vycor. The gases then flow into an ordinary glass receiving flask for  $TiCl_4$  and absorption tubes for PH<sub>3</sub> (one contains aqueous CuSO<sub>4</sub> and the other copper wire mesh).

The system is first filled with  $H_2$ . The reaction tube is heated to 750°C and the  $PH_3$  generator is started (this requires 3-4 hours). The TiCl<sub>4</sub>, which is kept cold up to this time, is now heated.

1242

The chloride vapor reacts with the phosphine in the hot reaction tube (reaction time: about three hours). As in the preparation of  $TiS_2$  from  $TiCl_4$  and  $H_2S$ , the  $TiCl_4$  may be cycled back and forth through the tube. The titanium phosphide product is a light-gray, high-polish mirror deposited on the tube walls. However, the product is not free of the chloride even after treatment at  $350^{\circ}C$  in high vacuum. Preparations with maximum phosphorus content correspond to  $TiP_{0.93}$ ; the yield is modest.

PROPERTIES:

Black-gray powder with a metallic appearance; attacked only slightly by acids even when heated; thermally very stable; it is assumed that a subphosphide exists. d 3.94.

**REFERENCES**:

W. Biltz, A. Rink and F. Wiechmann. Z. anorg. allg. Chem. <u>238</u>, 395 (1938); I. Gewecke. Liebigs Ann. <u>361</u>, 70 (1908); for data on the system TiCl<sub>4</sub>/PH<sub>3</sub>, see R. Höltje, Z. anorg. allg. Chem. <u>190</u>, 246 (1930).

# Zirconium Phosphides

The pressure synthesis used for TiP is also employed in the preparation of compounds of the Zr/P system. First stage: treatment in a Faraday tube for 50 hours at a 1000/500°C gradient; atomic ratio 1 Zr:4 P. Second stage: aftertreatment in a quartz pressure tube (wall thickness 4 mm., I.D. 12 mm.) for 50 hours at 800°C. The product is black-gray  $ZrP_2$ ; this may be degraded to ZrP in high vacuum at 820°C. If the desired composition does not exceed  $ZrP_{\leq 1}$ , the second stage merely increases the homogeneity of the product. Since vapor pressure then ceases to be a factor, ordinary quartz tubes may be used.

The diphosphide  $ZrP_2$  may also be prepared via method II for TiP.

REFERENCE:

# E. F. Strotzer and W. Biltz. Z. anorg. allg. Chem. 239, 216 (1938).

# Thorium Phosphide

One heating cycle in the Faraday apparatus suffices to prepare  $Th_3P_4$  (a subphosphide  $ThP_{0.6}$  also exists). The reactants are heated for 60 hours in a furnace with a 940/450°C gradient

(atomic ratio of Th:P = 1:3). The absorptivity of Th for x-rays is very high, and therefore minute quantities of surface oxide interfere in the x-ray pattern analysis. Therefore, for precision work (particularly when the product is a lower phosphide) an empty cylindrical crucible is first heated for one hour (under high vacuum) in the rear section of the tubular quartz reactor, while the small quartz flask with the raw material at the front end of the reactor remains cold. After cooling of the rear section (in high vacuum), the raw material is transferred to the crucible and the reactor tube is sealed off.

REFERENCE:

E. F. Strotzer and W. Biltz. Z. anorg. allg. Chem. 238, 69 (1938).

# Zirconium and Hafnium Phosphates

No unequivocal characterization exists for the phosphates precipitated from Zr salt solutions. The orthophosphate is formulated as either  $Zr(HPO_4)_2$  or  $ZrO(H_2PO_4)_2$ ; on prolonged heating above 700°C, it converts to  $ZrP_2O_7$ .

I. 
$$ZrOCl_2 \cdot 8 H_2O \rightarrow ZrO(H_2PO_4)_2 \rightarrow ZrP_2O_7$$
  
322.3 301.2 265.2

First,  $ZrOCl_2 \cdot 8H_2O$  (2 g.) is dissolved in 1.5 liters of 6N HCl, and then a solution of 2 g. of  $Na_2HPO_4$  in 1.5 liters of 6N HCl is added in drops. The finely divided precipitate is washed by repeated decantation with 6N HCl, filtered and dried at 80°C. The product corresponds to  $ZrO(H_2PO_4)_2$ .

The hafnium analogue,  $HfO(H_2PO_4)_2$ , is prepared in exactly the same manner; it is less soluble than the Zr salt.

II. Alternate method: Solutions of Zr sulfate  $(2-5\% \text{ ZrO}_2)$  and  $2.5\% \text{ H}_3\text{PO}_4$  in 2N sulfuric acid are added simultaneously in drops to 2N H<sub>2</sub>SO<sub>4</sub> at 70-75°C.

PROPERTIES:

Sparingly soluble in conc. mineral acids, except hydrofluoric acid; when freshly precipitated, soluble (with formation of a complex) in a mixture of  $H_3PO_4$ , oxalic acid and conc.  $H_2SO_4$ . An acid-soluble, white peroxy compound is formed when a cold suspension of the phosphate is reacted with an NaOH-Na<sub>2</sub>O<sub>2</sub> solution and then digested at 70°C.  $ZrP_2O_7$  crystallizes in the cubic K 6<sub>1</sub> lattice.

1244

**REFERENCES**:

- I. G. Hevesy and K. Kimura. Angew. Chem. <u>38</u>, 775 (1925); J. Amer. Chem. Soc. <u>47</u>, 2540 (1925); see also J. H. de Boer, Z. anorg. allg. Chem. <u>144</u>, 193 (1925).
- II. E. M. Larsen, W. C. Fernelius and L. L. Quill. Analyt. Chem. <u>15</u>, 512 (1943).

### Titanium, Zirconium and Hafnium Carbides

#### TiC, ZrC, HfC

# Titanium Carbide

Methods I to III, given in detail for the preparation of TiN, are also useful for synthesis of carbides. These are: I) the industrial process  $TiO_2 + C$ ; II) synthesis from the elements; and III) vapor deposition. The last process may be modified by first depositing the metal from a vapor and then converting the deposit (on a glow wire) to the carbide by heating in a hydrocarbon atmosphere (method IV), or by using a glowing carbon wire in an atmosphere of  $TiCl_4$  vapor (method V).

I.

$$TiO_2 + 3C = TiC + 2CO$$
  
79.9 36.0 59.9 56.0

In the method of Agte and Moers, a mixture of pure TiO<sub>2</sub> and acetylene-derived carbon black (the latter degassed at 2000 °C) is placed in a graphite boat and heated for half an hour in a tubular graphite furnace to 1700-1800 °C. Very pure and dry H<sub>2</sub> is used to flush the apparatus. Since the hydrogen reacts with the hot graphite tube to form hydrocarbons, the carbon content of the raw material mixture should be 15-25% less than the stoichiometric ratio. The products usually still contain some oxygen.

If the  $TiO_2 + 3C$  mixture is heated very rapidly (within 20 minutes) to 1900°C in a stream of H<sub>2</sub> or CO, a product containing 19.5% C may be obtained (as demonstrated by Meyerson); further heating reduces the carbon content to 17%, because of decarbonization.

#### PRESINTERING

The carbide rods, made of powder compressed at 2000 kg./cm<sup>2</sup>, are heated in a graphite boat inside a tubular graphite furnace to temperatures between 2500 and 3000°C and maintained at these temperatures for about 15 minutes. The material is protected against surface absorption of additional carbon by embedding the rods in carbide powder.

Due to considerable shrinkage of the rods, the processes of repulverization, recompression and resintering must be repeated 2-4 times.

#### HIGH-TEMPERATURE SINTERING

Since the carbides decompose in vacuum, the high-temperature sintering must be carried out in an atmosphere of technicalgrade argon. The rods are heated to extremely high temperatures just short of the melting point (for a description of this procedure, see TiN, p. 1234). Most of the impurities evaporate at these temperatures, leaving a relatively pure product.

II.	Ti +	$\mathbf{C} =$	TiC
	47.9	12.0	59.9

Titanium filings are mixed in the stoichiometric ratio with acetylene-derived carbon black (very thoroughly degassed at 2000°C), and the reaction is started by heating to 1800°C in a BeO boat placed inside a high-vacuum furnace (see Part I, p. 40 for description). The beryllium oxide boats are set up inside the heating element, which consists of tungsten boats (40 mm. long, 10 mm. wide, 8 mm. high) subjected to high-intensity (200 amp.) low-voltage current. The final sintering of the finely powdered crude product requires 10 minutes at 2400°C.

III. In this process a tungsten wire, which serves only as a substratum for the deposit, is heated to glowing in an atmosphere consisting of a volatile halide of the metal, a carbon compound and  $H_2$ . Moers recommends the use of hydrocarbons such as toluene, instead of CO; the deposition of free carbon with the carbide is avoided if the partial pressure of the hydrocarbon in the system is low. The hydrogen atmosphere facilitates considerably the reaction at the glow wire by reducing the decomposition temperatures of the halides to a much greater extent than does reduced pressure or even vacuum.

 $TiCl_4 + CH_4 \stackrel{(+H_2)}{=} TiC + 4 HCl$ 189.7 16.0 59.9 145.8

A pure hydrogen stream (which must be free of  $N_2$  and  $O_2$  and is therefore most conveniently generated by electrolysis) is divided into two fractions, one of which is passed through a 25 °C wash bottle filled with TiCl<sub>4</sub>, the other through a similar bottle containing toluene at -15 °C; the streams are then recombined and introduced into the reactor. The glow wire is maintained at a temperature of 1600 °C, which is kept constant during the experiment by gradually increasing the current. Further details are given by K. Moers.

1246

### Alternate methods:

IV. A modification of the method just described consists in heating the metal wire prepared by the vapor deposition process in a hydrocarbon atmosphere. This modification is not, however, very convenient in the case of the lower-melting metals (Ti, Zr) since the wire temperatures must be relatively low and thus very long glow times are required. It may be used successfully with W, Ta and Hf.

V. Another method of avoiding introduction of the tungsten wire (which is used as a substratum in all previously described vapor deposition processes) into the product reverses the above procedure; i.e., a carbon wire is heated to incandescence in the vapor of a volatile halide of the metal (in the presence or in the absence of  $H_2$ ).

The method suffers from one disadvantage: the dissociation of the chlorides at the glow wire does not cease when all the carbon originally present in the wire has been consumed. As a result, the products contain varying amounts of the free metal dissolved in the carbide. Carbides of stoichiometric composition are obtained either by calcining the above products in high vacuum (to evaporate the dissolved metal) or in a hydrocarbon atmosphere (to convert the excess metal to carbide). Further details are given in the reference cited below.

#### PROPERTIES:

Gray powder. Insoluble in hydrochloric acid, soluble in nitric acid. M.p. 3410°C; d 4.92. Very good electrical conductor with a positive temperature coefficient. Crystal structure: type B1 (NaCl type), with a considerable range of phase compositions (TiC<sub>1.0</sub> -  $Ti_{0.3}$ ).

# Zirconium Carbide and Hafnium Carbide

These compounds are prepared by the same procedures as titanium carbide. The reaction mixture consists of either  $\text{ZrO}_2$  (or  $\text{HfO}_2$ ) + 3 C, or Zr (or Hf) + C; the reaction temperatures lie above 2000°C. Since both ZrC and HfC are very sensitive to N<sub>2</sub>, the high-temperature sintering stage must be carried out in 99% Ar.

Just like TiC, both ZrC and HfC dissolve carbon when molten. This phenomenon is most detrimental in the case of ZrC, whose melting point is lowered from 3500°C to about 2450°C by the absorption of carbon; the carbon is released on cooling.

The procedures employed for the preparation of TiC by the vapor deposition process must be modified somewhat in the case of ZrC and HfC, since both  $ZrCl_4$  and HfCl<sub>4</sub> are solids. The

reactor is filled with a sufficient quantity of the chloride and its lower section is heated with a small furnace to a temperature at which the vapor pressure is about 10-20 mm. (that is, to about  $300^{\circ}$ C). The same applies in method V.

The industrial preparation of ZrC from  $ZrSiO_4$  and carbon in a graphite crucible using an arc furnace is described by W. Kroll et al. [Trans. Electrochem. Soc. <u>89</u>, 263, 317 (1946); <u>92</u>, 187 (1947); J. Electrochem. Soc. <u>94</u>, 1 (1948)].

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# Thorium Carbides

### ThC, ThC<sub>2</sub>

Thorium carbides are prepared in an electric arc furnace. The arc is produced in a graphite crucible containing the reaction mixture. About 200 amp. at 110 v. is required to melt the mixture.

A mixture of  $ThO_2$  and calcined carbon black (0.24% ash) or graphite powder (0.33% ash), in quantities corresponding to the above equations, is made into a paste with a small amount of water and starch, and evaporated with stirring (graphite is preferred to carbon black because of its smaller volume). The lumps of dried material are placed in a crucible by means of a porcelain spatula; their large size prevents them from being carried out of the crucible by the CO gas evolved in the process.

PROPERTIES:

ThC  $_2$ : Opaque, dark-yellow pseudotetragonal crystals with a metallic luster. M.p. 2650°C; d 8.96. Completely miscible with

ThC at high temperature, practically immiscible at room temperature. Forms a eutectic with graphite (empirical formula  $ThC_{2.8}$ , m.p. 2500°C). Decomposes slowly in water, rapidly indilute acids, evolving a mixture of hydrocarbons (chiefly acetylene) and H<sub>2</sub>.

ThC: M.p. 2620°C. Crystal structure: NaCl type. For material of empirical formula  $ThC_{0,39}$ , the miscibility gap begins at a temperature of 1980°C; the gap widens at room temperature to  $ThC_{0,05}$ -ThC<sub>0,76</sub>.

**REFERENCES**:

M. von Stackelberg. Z. phys. Chem. (B) <u>9</u>, 437 (1930); H. A. Wilhelm and P. Chiotti. Trans. Amer. Soc. Metals <u>42</u>, 1295 (1950).

# Titanium, Zirconium and Thorium Silicides

### TiSi<sub>2</sub>, ZrSi<sub>2</sub>, ThSi<sub>2</sub>

### I. PREPARATION FROM THE ELEMENTS

 $\begin{array}{rll} {\rm Ti}\,({\rm Zr},{\rm Th}) \ + \ 2\,{\rm Si} \ = \ {\rm TiSi}_2\,({\rm ZrSi}_2,\ {\rm ThSi}_2) \\ {\rm 47.9\,91.2}\ 232.1 \ \ 56.1 \ \ 104.0 \ \ 147.3 \ \ 288.2 \end{array}$ 

The silicides are prepared by fusing or sintering intimate stoichiometric mixtures of the elements (in powder form). The reaction is carried out in sintered clay or ceramic crucibles, placed in a Tammann furnace under a blanket of Ar. In all three cases the reaction takes place at a relatively low temperature and is highly exothermic; therefore, one should work with gram quantities only.

In the method of Alexander, some advantage is gained by replacing the pure metals with hydride powders. Initial heating takes place in vacuum, which at 400-500 °C is replaced by an atmosphere of the  $H_2$  evolved by the reaction itself. This gives a sintered silicide at temperatures as low as 900 °C.

Brauer and Mitius modified a method developed by Hönigschmid and prepare ThSi<sub>2</sub> by the following procedure. Intimate mixtures of Al, Th and Silumin (13% Si, 87% Al) powders are compressed into tablets 10 mm. long and 5 mm. thick, placed in alumina crucibles and fused at 1100 °C in high vacuum. Slow cooling (from 1100 °C to 800 °C in four hours) produces good crystals of the product within the aluminum ingot. These are freed of excess Al by alternate treatments with dilute hydrochloric acid and potassium hydroxide (moderate heating), followed by washing with water and alcohol. Since most of the silicide particles form cohesive
granules with the Si and  $\text{SiO}_2$  contaminants, the crystallites are pulverized in an agate mortar and separated from foreign matter (particularly  $\text{SiO}_2$ , which cannot be removed by chemical means) by flotation with bromoform (d 2.9). The bromoform must be continuously renewed. Evaporation with hydrofluoric acid is not practical, since it destroys the silicide.

Alternate methods:

# II. ALUMINOTHERMIC METHOD

Ignition of a mixture of, for example, 200 g. of Al powder, 250 g. of S, 180 g. of SiO<sub>2</sub> and 15 g. of TiO<sub>2</sub> (or 40 g. of  $K_2TiF_6$ ), covered with a thin layer of Mg, yields an ingot containing, in addition to Si and Al, small, iron-gray tetragonal pyramids of TiSi<sub>2</sub>.

### III. ELECTROLYTIC PREPARATION

The pure, crystalline silicides are obtained by melt electrolysis of a mixture of, for example,  $10 \text{ K}_2 \text{SiF}_6 + \text{TiO}_2$  at about 900°C, using an iron cathode; alternately, electrolysis of TiO<sub>2</sub> dissolved in a melt of silicate may be used.

IV. REACTION OF THE METAL WITH SILICON TETRACHLORIDE

The silicides are obtained in the form of a coating on the reactor walls when the metals are heated to 1100-1500 °C in a hydrogen stream saturated with SiCl<sub>4</sub>.

PROPERTIES:

Grayish-white crystals with a metallic luster and good thermal and electrical conductivity. d  $(TiSi_2) 4.02$ ;  $(ZrSi_2) 4.88$ ;  $(ThSi_2)$ 7.63. TiSi<sub>2</sub> and ZrSi<sub>2</sub> are insoluble in mineral acids (except hydrofluoric acid); ZrSi<sub>2</sub> is also insoluble in 10% KOH. TiSi<sub>2</sub> dissolves slowly in 10% KOH. ThSi<sub>2</sub> is unaffected by alkali, but dissolves in dilute and conc. hydrogen halides (slowly in the cold and rapidly when heated). Attacked by Cl<sub>2</sub> at temperatures as low as 500°C. TiSi<sub>2</sub> is stable at red heat in air, while ZrSi<sub>2</sub> and ThSi<sub>2</sub> burn.

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I. O. Hönigschmid. Monatsh. Chemie <u>28</u>, 1017 (1907); P. P. Alexander. Metals and Alloys <u>9</u>, 179 (1938); F. Laves and H. J. Wallbaum. Z. Kristallogr. <u>101</u>, 78 (1939); G. Brauer and A. Mitius. Z. anorg. allg. Chem. <u>249</u>, 325 (1942); E. L. Jacobson, R. D. Freeman, A. G. Tharp and A. E. Searcy. J. Amer. Chem. Soc. <u>78</u>, 4850 (1956); see also H. J. Wallbaum. Z. Metallkunde 33, 378 (1941).

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# **SECTION 23**

# Vanadium, Niobium, Tantalum

G. BRAUER

### Vanadium

v

The preparation of high-purity V metal is difficult because it tends to form very stable occluded phases with nonmetals, particularly with O, N and C. These elements must be removed in advance, because their elimination from the metal phase at a later stage is difficult. Hence only a few of the many proposed methods afford a ductile metal or V powder of a high degree of purity.

I.

 $V_2O_5 + 5Ca = 2V + 5CaO$ 181.9 200.4 101.9 280.4

In the method of Marden and Rich, 175 g. of  $V_2O_5$  is mixed with 300 g. of ground Ca and 300 g. of CaCl<sub>2</sub> (dehydrated by preheating at 450°C), and the mixture placed in a small sealable iron bomb. The addition of CaCl<sub>2</sub> as fluxing agent is essential. A small piece of Na or K is also added (to act later as a scavenger for residual  $O_2$  and  $H_2O$ ), or the bomb is evacuated and then filled with Ar. The bomb is tightly sealed either by screwing down the lid or by welding, heated one hour at 900-950°C, cooled to room temperature and then reopened. The product is chipped out with a chisel, and the chunks are added, slowly and with agitation, to about 20 liters of cold water (avoid local overheating). The product disintegrates and is allowed to settle for about 2 min. The supernatant is decanted; the solid is washed (by decantation) several times with H<sub>2</sub>O, then several times with approximately 2 N HCl. The product is ductile, granular V metal.

According to McKechnie and Seybolt, the reaction between  $V_2O_5$  and Ca is coupled to advantage with the strongly exothermic reaction Ca +  $I_2 = CaI_2$ . In this case, CaI<sub>2</sub> has the desirable effect of reducing the melting point of the mixture, thus replacing the CaCl<sub>2</sub> used in the previous method.

For example, a mixture of 300 g. of specially purified  $V_2O_5$  (see below), 552 g. of metallic Ca (i.e., about 60% excess) and 150 g. of

 $I_2$  is placed in a sintered magnesia crucible, which is in turn placed in a 1.5-liter steel bomb; the bomb is then hermetically sealed. With the above quantities, the steel bomb should have a diameter of 100 mm., a height of 280 mm. and a 10-mm.-thick wall; the lid should be a steel plate held between bolted-down flanges. The magnesia crucible can be closed with a cover of the same material (the cover is formed by pressure-shaping granular magnesia with a steel cover). The bomb is evacuated, filled with Ar, closed and heated to about 425°C to start the reaction. Immediately after ignition, spontaneous heating to a much higher temperature takes place, accompanied by an unusual noise. Granular and powdered V is isolated from the reaction product in 74% yield.

The  $V_2O_5$ , which is normally prepared from NH<sub>4</sub>VO<sub>3</sub>, must be freed before use of small amounts of N and H still adhering to it. One method of accomplishing this is heating the oxide for 18 hours in a stream of moist  $O_2$  at 400°C.

II. The following reduction affords V as a fine powder:

$$VCl_3 + 1^{1/2}H_2 = V + 3 HCl$$
  
157.3 33.6 l. 51.0 109.4

A stream of  $H_2$ , thoroughly freed of traces of  $O_2$ ,  $N_2$  or  $H_2O$ (see p. 112 ff.,  $N_2$  removal as for Ar, p. 82 ff.), is passed over a platinum boat located in a Pt tube and containing about 7 g. of VCl<sub>3</sub>. The Pt tube forms an insert for a porcelain tube and protects the latter from attack by the subliming VCl<sub>3</sub> and by the V formed from such a sublimate; it also ensures the protection of the boat contents from contamination by Si compounds from the porcelain tube. The V formed during the reaction is further protected against contamination by placing a porcelain boat containing powdered V ahead of the Pt boat. This protective vanadium (which may be less pure) serves to remove the last traces of  $N_2$  and  $O_2$  from the gas stream. The reactor tube is connected to a large, empty U tube which allows observation of the exit gas and is, in turn, connected to a KOH-filled trap.

The porcelain tube is slowly heated to, and then held at, a temperature of 900°C until HCl evolution (which follows the initial formation of a small quantity of brown fumes) is complete. After cooling to room temperature (and not before), the product is removed from the reactor in a stream of H<sub>2</sub>. At this point, it consists of vanadium hydride (approximate composition  $VH_{1.7}$ ), which is converted to pure V by heating in high vacuum. It should be borne in mind that finely subdivided vanadium and vanadium hydride are sensitive to atmospheric O<sub>2</sub> even at room temperature.

The Pt sheet absorbs some V during the reaction and turns darker, brittle and fragile. The vanadium can be removed from the Pt in the form of  $V_2O_5$  by heating to red heat in air or, via a more drastic method, by treatment with a molten mixture of 1 part of KNO<sub>3</sub> and 15 parts of NaKCO<sub>3</sub>. The Pt is thus completely regenerated. It should be possible to replace the porcelain tube, with no loss of efficiency, by an alumina tube.

Alternate methods: a) Very pure V can be obtained by deposition from the gas phase in the apparatus shown in Fig. 291, p. 1168. In this case the apparatus is made of fused quartz and is heated to 900-1000°C during the reaction. A suitable crude metal starting product is obtained, for example, by reacting a mixture of VCl<sub>3</sub> and Na in a heated iron bomb. Since the transport to the incandescent wire is accomplished via VI<sub>2</sub>, whose volatility is relatively low, this process is not as advantageous in the case of V as with metals of Group IV [A. E. van Arkel, Metallwirtsch. <u>13</u>, 405 (1934); J. W. Nash, H. R. Ogden, R. E. Durtschi and I. E. Campbell, J. Electrochem. Soc. <u>100</u>, 272 (1953); H. W. Rathmann and H. R. Grady, Vancoram Rev. <u>10</u>, 6, 17 (1955)].

b) With Ar as the carrier phase, the reduction of VCl<sub>4</sub> with  $H_2$  (affording pure V powder) can be accomplished at 620°C [G. Jantsch and F. Zenek, Monatsh. Chem. 84, 1119 (1953)].

c) According to another proposed method, a stream of dry, high-purity H<sub>2</sub> saturated with VCl<sub>4</sub>vapor is passed over Mg turnings (in a MgO boat) and gradually heated to 700 °C over a period of 2.5 hours. After cooling, the product mixture of V, VCl<sub>2</sub> and VCl<sub>3</sub> is thoroughly extracted with H<sub>2</sub>O to dissolve out any chlorides present. The residue of V powder (99.3% V) is then vacuum-dried.

d) A mixture of 2 parts of VCl<sub>2</sub> and 1 part of Mg is pressed into pellets and these are heated for 1-2.5 hours at 700°C in a MgO boat inserted in a quartz tube (H<sub>2</sub> or Ar atmosphere). The product metal contains up to 99.5% V [A. Morette, Comptes Rendus Hebd. Séances Acad. Sci. 200, 1110 (1935)].

Solid V metal is comminuted and reduced to a fine powder via the vanadium hydride stage. Thus, vanadium is heated to about 500 °C in a stream of high-purity H<sub>2</sub> and is then cooled in the same gas. The hydride is very brittle and can be readily comminuted by pounding or grinding in a volatile organic liquid such as benzene, which protects it against local heating and oxidation. The comminuted material is then dehydrogenated to pure V metal by heating in high vacuum (W. D. Schnell, Thesis, Univ. of Freiburg i. Br., 1960).

The purity of high-grade V can be further increased (to about 99.997%) by long heating (e.g., for 20 hours) in a high vacuum  $(10^{-5} \text{ mm.})$  at 1650°C (W. D. Schnell, Thesis, Univ. of Freiburg i. Br., 1960).

For information concerning melting V metal in crucibles made of various materials, see T. W. Merril, Vancoram Rev. <u>11</u>, 11, 16 (1956). PROPERTIES:

Atomic weight 50.95. Light-gray metal, ductile when pure. M.p. 1900°C; d 6.11. Insoluble in hydrochloric and sulfuric acids, soluble in nitric and hydrofluoric acids. High affinity for O, N, C and H. Surface reaction with atmospheric  $O_2$  starts already at 20°C; this can lead, particularly in the case of a fine powder, to considerable contamination. Crystal structure: type  $A_2$ .

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# Vanadium (II) Chloride VCI<sub>2</sub>

I.

 $VCl_3 + \frac{1}{2}H_2 = VCl_2 + HCl_{157.3} + \frac{11.2}{11.2}l_{121.9} + \frac{1121.9}{36.5}$ 

The reactor is a Pyrex, Vycor or fused quartz tube, and the VCl<sub>3</sub> is placed either directly in the tube or in a porcelain boat (the transfer to the reactor requires great care, because VCl<sub>3</sub> is very hygroscopic). A stream of dry, completely deoxygenated hydrogen is then passed through the tube (the end of the tube is protected against moisture by a CaCl<sub>2</sub> tube), and the system is heated to about 400°C. While HCl is evolved, the temperature is gradually increased to 675°C. Care should be taken not to exceed 700°C (according to Klemm and Hoschek, not even 500°C), otherwise reduction to V metal will occur. The reaction time depends on the quantity of material used; it is about 1 hour for 0.5 g., and about 40 hours for 30 g. At the end of the reaction, the H<sub>2</sub> stream is replaced by a stream of  $P_2O_5$ -dried  $N_2$  or CO<sub>2</sub>, and the VCl<sub>2</sub> produced is discharged from the reactor under anhydrous conditions. A high yield (~90%) is obtained.

II. 
$$2 \text{VCl}_3 = \text{VCl}_2 + \text{VCl}_4$$
  
314.6 121.9 192.8

Rapid disproportionation of  $VCl_3$  according to the above equation can be achieved at 800°C in a stream of high-purity N<sub>2</sub>. The  $VCl_4$  is carried away by the  $N_2$  stream (4 bubbles/sec.), while the VCl<sub>2</sub> remains in the reactor tube. The temperature should not exceed 850°C, to avoid loss of VCl<sub>2</sub> by sublimation. The reaction is fairly fast; for example, the reaction of 20 g. of VCl<sub>3</sub> takes 2 hours.

PROPERTIES:

Light-green leaflets; m.p. about  $1350^{\circ}$ C; d. 3.09. Less hygroscopic than VCl<sub>3</sub> and VCl<sub>4</sub>; insoluble in alcohol or ether. Crystal structure: CdI<sub>2</sub> type.

**REFERENCES**:

- F. Ephraim and E. Ammann. Helv. Chim. Acta <u>16</u>, 1273 (1933);
   W. Klemm and E. Hoschek. Z. anorg. allg. Chem. <u>226</u>, 359 (1936);
   R. C. Young and M. E. Smith in: J. C. Bailar, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 126;
   H. Funk and W. Weiss. Z. anorg. allg. Chem. <u>295</u>, 327 (1958).
- II. O. Rugg and H. Lickfett. Ber. dtsch. chem. Ges. <u>44</u>, 506 (1911);
   P. Ehrlich and H. J. Seifert. Z. anorg. allg. Chem. <u>301</u>, 282 (1959).

# Vanadium (III) Chloride

VCI<sub>3</sub>, VCI<sub>3</sub> · 6 H<sub>2</sub>O

VCI<sub>3</sub>

I.

 $V_2O_3 + 3 SOCl_2 = 2 VCl_3 + 3 SO_2$ 149.9 356.9 314.6 192.2

Vanadium trioxide powder (2.1 g.) and 8.5 ml. of pure SOCl<sub>2</sub> are placed in a bomb tube about 1.5 cm. in diameter, and the sealed tube is heated for 24 hours at 200°C. The tube is cooled to below 0°C (in order to lower the SO<sub>2</sub> vapor pressure) and is then opened to allow SO<sub>2</sub> to escape. Then the tube contents are flushed out, under anhydrous conditions, into a small flask, using some SOCl<sub>2</sub> for this purpose (the SOCl<sub>2</sub> is then removed by vacuum distillation). The VCl<sub>3</sub> residue is washed several times with very pure CS<sub>2</sub> to remove traces of S<sub>2</sub>Cl<sub>2</sub>, and then thoroughly vacuum-dried at 80°C. Fine crystals of dark-violet VCl<sub>3</sub> are obtained in nearly quantitative yield.

II.

Fine, pure  $V_2O_5$  powder (18 g.) and 40 ml.ofS<sub>2</sub>Cl<sub>2</sub> are refluxed under anhydrous conditions for 8 hours (constant stirring). The excess  $S_2Cl_2$ , containing dissolved S, is decanted and the VCl<sub>3</sub> formed is washed with dry CS<sub>2</sub>. Adhering volatiles are removed by heating the material at 120-150°C under vacuum or by extracting it for several hours with CS<sub>2</sub> in a Soxhlet apparatus. After thorough purification, the residual sulfur content of the resulting fine crystals of VCl<sub>3</sub> is about 0.2%. The yield is about 30 g.

Coarse (and hence less hygroscopic) crystals of VCl<sub>3</sub> are obtained by heating the fine crystalline product with a small amount of fresh  $S_2Cl_2$  in a sealed tube at 240°C. Since no gas is evolved in this operation, large amounts can be treated at one time.

The same reaction can also be carried out in a sealed tube at  $300^{\circ}$ C; however, smaller quantities must be used in this case (6-7 g. of  $V_{2}O_{5}$  and 20 ml. of  $S_{2}Cl_{2}$ ).

III.  $VCl_4 = VCl_3 + \frac{1}{2}Cl_2$ 192.8 157.3 35.5

A flask containing VCl<sub>4</sub> is connected to a reflux condenser via a ground joint and kept at 160-170 °C for 2 days while passing a thoroughly dried and deoxygenated stream of CO<sub>2</sub> or H<sub>2</sub> through the system (to remove the Cl<sub>2</sub> formed in the thermal dissociation). The flask is then arranged for distillation; unreacted VCl<sub>4</sub> and the traces of the VOCl<sub>3</sub> formed are distilled off at 200°C in a H<sub>2</sub> stream. Reduction does not take place under these conditions. The by-products can also be removed by vacuum distillation. All operations are carried out under anhydrous conditions.

Alternate methods: IV. Reaction of VCl<sub>4</sub> with S, removal of the  $S_2Cl_2$  by distillation, then heating of the product at a temperature somewhat below 300°C in a stream of CO<sub>2</sub>.

V. If V metal is available as a very fine powder, it can be reacted with excess ICl, according to the equation:

$$2V + 6ICl = 3I_2 + 2VCl_3$$
  
101.9 974.2 761.5 314.6

The mixture is heated under anhydrous conditions in a glass flask equipped with a reflux condenser. The flask is carefully heated with a direct flame until the initially vigorous reaction, which affords iodine vapor, subsides and vapors of the boiling ICl become visible. After cooling, the mixture is extracted with  $CCl_4$  (distilled from  $P_2O_5$ ), filtered in a  $N_2$  stream through a sintered glass disk, washed with  $CCl_4$  and dried in a vacuum desiccator. The product is very pure provided all of the vanadium has reacted.

PROPERTIES:

Formula weight 157.3. Violet, quite crystalline. Very hygroscopic. Soluble in acidified water. In the absence of air, can be obtained from solutions as  $VCl_3 \cdot 6H_2O$ . Soluble in alcohol, insoluble in ethyl ether. d 3.0. Crystal structure:  $DO_5$  (FeCl<sub>3</sub>) type.

REFERENCES:

- H. Hecht, G. Jander and H. Schlapmann. Z. anorg. Chem. <u>254</u>, 255 (1947); experiments carried out at the University Laboratory, Freiburg i. Br., 1951.
- II. H. Funk and C. Müller. Z. anorg. allg. Chem. <u>244</u>, 94 (1940); experiments carried out at the University Laboratory, Freiburg
  i. Br., 1951; H. Hartmann and H. L. Schläfer. Z. Naturforsch. <u>6a</u>, 754 (1951); H. Funk and W. Weiss. Z. anorg. allg. Chem. <u>295</u>, 327 (1958).
- III. J. Meyer and R. Backa. Z. anorg. allg. Chem. <u>135</u>, 177 (1924);
  F. Ephraim and E. Ammann. Helv. Chim. Acta <u>16</u>, 1273 (1933);
  R. C. Young and M. E. Smith in: J. C. Bailar, Inorg. Syntheses, Vol. IV, New York-Toronto-London, 1953, p. 128.
- IV. O. Ruff and H. Lickfett. Ber. dtsch. chem. Ges. 44, 506 (1911);
   F. Ephraim and E. Ammann. Helv. Chim. Acta 16, 1273 (1933).
- V. V. Gutmann. Monatsh. Chem. 81, 1155 (1950).

VCl<sub>3</sub> · 6 H<sub>2</sub>O

The hexahydrate can be obtained from aqueous, acidic  $VCl_3$  solution by cooling and saturating with HCl. The starting solution is obtained by electrolytic reduction of a solution of  $V_2O_5$  in hydrochloric acid or, more conveniently, by dissolving  $V_2O_3$  in hydrochloric acid.

 $\begin{array}{rrrr} V_2O_5 &+& 2\ H_2 &=& V_2O_3 &+& 2\ H_2O\\ 181.9 & 44.8\ l. & 149.9 \end{array}$  $\begin{array}{rrrr} V_2O_3 &+& 6\ HCl &+& 9\ H_2O &=& 2\ VCl_3\cdot 6\ H_2O\\ 149.9 & 134.4\ l. & 162.1 & 265.4 \end{array}$ 

For example, 7.5 g. of  $V_2O_3$  (obtained by reduction of  $V_2O_5$  with  $H_2$  as described on p. 1267) is dissolved in 200 ml. of conc. HCl by allowing the mixture to boil several hours. This solution is concentrated to 50 ml., cooled to -10 to -20 °C, and saturated with HCl gas. The precipitate of green VCl<sub>3</sub> · 6 H<sub>2</sub>O is suction-filtered on glass frit, dissolved in some  $H_2O$ , and reprecipitated with HCl while cooling (see also the preparation of TiCl<sub>3</sub> · 6 H<sub>2</sub>O, p. 1193 f.).

PROPERTIES:

Formula weight 265.4. Green, hygroscopic crystals.

REFERENCES:

A. Piccini and M. Brizzi. Z. anorg. Chem. <u>19</u>, 394 (1899); P. Ehrlich and H. J. Seifert. Z. anorg. allg. Chem. 301, 282 (1959).

### Vanadium (IV) Chloride

#### $VCI_4$

I.

$VCl_3$	+	$1/_2 \operatorname{Cl}_2$	=	VCl <sub>4</sub>
157.3		11.2 l.		192,8

In the method of Funk and Weiss VCl<sub>3</sub>, which is readily obtained from  $V_2O_5$  and  $S_2Cl_2$  (see p. 1256), is loosely packed in a slightly inclined reactor tube made of high-melting glass. The tube is connected by a ground joint to a receiver, which is protected against moisture. A glass-wool wad is placed at the end of the reactor tube to prevent solid VCl<sub>3</sub> particles from reaching the receiver. Dry  $Cl_2$  is passed through the apparatus to displace the air; the VCl<sub>3</sub> is then heated, starting from the end closest to the receiver. The reaction rate can be controlled by regulating the  $Cl_2$  flow. The crude VCl<sub>4</sub> is collected in the receiver, which is kept at 0°C; it is then distilled from this receiver in a slow stream of  $Cl_2$  (anhydrous conditions), discarding the forerun. Approximately 35 g. of pure VCl<sub>4</sub> is obtained from 30 g. of VCl<sub>3</sub>.

II. A process similar to that described in method I may be used to prepare  $VCl_4$  from ferrovanadium and  $Cl_2$ . A very long reactor tube is used, and the reaction rate is regulated so as to allow most of the by-product FeCl<sub>3</sub> to settle out. The VCl<sub>4</sub> must be redistilled from the receiver.

Alternate method: Disproportionation of VCl<sub>3</sub> to VCl<sub>2</sub> and VCl<sub>4</sub> by heating in a N<sub>2</sub> stream at 900°C in a porcelain tube [O. Ruff and H. Lickfett, Ber. dtsch. chem. Ges. <u>44</u>, 506 (1911)].

PROPERTIES:

Dark red-brown liquid. M.p. -109 °C, b.p. 148.5 °C; d 1.87. Fumes in air, and even at normal temperature shows a marked Cl<sub>2</sub> vapor pressure (decomposition). Sealed ampoules containing VCl<sub>4</sub> should be stored in the dark (occasionally they shatter because of high internal pressure). Decomposed by water. Soluble in conc. hydrochloric acid, ethanol and ethyl ether.

REFERENCES:

- I. H. Funk and W. Weiss. Z. anorg. allg. Chem. <u>295</u>, 327 (1958).
- II. A. T. Mertes. J. Amer. Chem. Soc. <u>35</u>, 671 (1913); F. Ephraim and E. Ammann. Helv. Chim. Acta <u>16</u>, 1273 (1933);
   J. H. Simons and M. G. Powell. J. Amer. Chem. Soc. <u>67</u>, 75 (1945).

# Vanadium (II) Bromide

### VBr<sub>2</sub>

 $VBr_3 + \frac{1}{2}H_2 = VBr_2 + HBr$ 290.7 11.2 l. 210.8 80.9

The starting material is  $VBr_3$ , which is best left in the tube used for its preparation (see below). In this case, the reactor tube is considerably longer than that used merely to prepare  $VBr_3$ .

The zones containing  $VBr_3$  are heated, one after another, to dark-red heat while a  $H_2$  stream is passed through. The heating is best accomplished in a tubular electric furnace, in which heating is more uniform than with open flames; with such a furnace, pronounced local overheating and reaction rate differentials are prevented and the total reduction time is short. The reduction of 2 g. of VBr<sub>3</sub> takes 1-1.5 hours.

PROPERTIES:

Light-brown to reddish; light pink-red when heated. Feltlike to flaky crystal aggregates. More hygroscopic than VCl<sub>2</sub>, but not as sensitive as VBr<sub>3</sub>. Gives a violet solution with  $H_2O$ ; this soon turns brown, evolving  $H_2$ . d 4.58. Crystal structure: C6 (CdI<sub>2</sub>) type.

**REFERENCES**:

F. Ephraim and E. Ammann. Helv. Chim. Acta <u>16</u>, 1273 (1933);
 W. Klemm and E. Hoschek. Z. anorg. allg. Chem. <u>226</u>, 359 (1936).

# Vanadium (III) Bromide VBr<sub>3</sub>

 $V(+Fe) + \frac{3}{2}Br_2 (+\frac{3}{2}Br_2) = VBr_3 (+FeBr_3)$ 51.0 55.9 239.7 239.7 290.7 295.6

A high-melting glass tube, 10-12 mm. I.D. and 80 cm. long, is charged with 5 g. of very finely powdered high-grade ferrovanadium (no boat is used). A stream of thoroughly dried CO<sub>2</sub> is passed through a small round-bottom flask in which absolutely dry  $Br_2$ , slowly introduced from a dropping funnel, is vaporized by slight heating. The CO<sub>2</sub> and  $Br_2$ vapors then pass through the reactor tube. The other end of the reactor carries (preferably sealed on) containers for the subsequent collection of VBr<sub>3</sub> (e.g., tubes which can be melt-sealed), which are protected from the atmosphere by a  $P_2O_5$  tube. Air and moisture must be rigorously excluded. After filling the reactor with  $CO_2 + Br_2$ , the Fe-V is heated to red heat. The first product is a small amount of VOBr<sub>3</sub>, which is quickly and readily displaced by heating the entire tube length to prevent the deposition of vanadium oxide (the latter cannot be removed). The conversion to VBr<sub>3</sub> and FeBr<sub>3</sub> is then carried out over a period of about 4 hours. Most of the FeBr<sub>3</sub> remains at the exit end of the tube, while the Br<sub>3</sub> sublimes out. It is freed of FeBr<sub>3</sub> by repeated sublimation. Rather low sublimation temperatures and long sublimation tubes are required in this procedure. Finally the system is flushed with pure, Br<sub>2</sub>-free CO<sub>2</sub>.

PROPERTIES:

Black with greenish reflections, crystalline; the vapor is violet. Extremely hygroscopic. d 4.52.

REFERENCES:

J. Meyer and R. Backa. Z. anorg. allg. Chem. <u>135</u>, 177 (1924);
 F. Ephraim and E. Amman. Helv. Chim. Acta <u>16</u>, 1273 (1933);
 W. Klemm and E. Hoschek. Z. anorg. allg. Chem. <u>226</u>, 359 (1936).

# Vanadium (II) Iodide

### VI2

This compound is prepared by synthesis from the elements.

 $V + I_2 = VI_2$ 51.0 253.8 304.8

I. In the method of Morette, VI<sub>3</sub> is first prepared from V and I<sub>2</sub> (see the following preparation) and then decomposed by heating at 400°C in high vacuum while removing the I<sub>2</sub> split off. The decomposition is virtually complete in 24 hours.

II. In the method of Klemm and Grimm, a stoichiometric mixture of V turnings and  $I_2$  is sealed under vacuum into a short quartz tube, with occasional cooling to reduce the  $I_2$  vapor pressure. The entire tube is then uniformly heated at 160-170°C.

PROPERTIES:

Dark-violet, hexagonal leaflets. Not readily wetted by  $H_2O$ ; slowly forms a violet solution. Insoluble in absolute ethanol, benzene, CCl<sub>4</sub>, CS<sub>2</sub>. Partly oxidized in air, turning brown. d 5.44. Crystal structure: C6 (CdI<sub>2</sub>) type. REFERENCES:

A. Morette. Comptes Rendus Hebd. Séances Acad. Sci. <u>207</u>, 1218 (1938); W. Klemm and L. Grimm. Z. anorg. allg. Chem. <u>249</u>, 198 (1942).

# Vanadium (III) Iodide

### VI<sub>3</sub>

Formed when metallic V and  $I_2$  are heated at 300°C under the vapor pressure generated by the latter.

 $\begin{array}{rcl} V &+& 3/_2 I_2 &=& V I_3 \\ 51.0 & 380.7 & 431.7 \end{array}$ 

Vanadium metal (turnings or powder) and excess  $I_2$  are placed in a hard glass or fused quartz tube, closed at one end; the tube contents are cooled to  $-80^{\circ}$ C; the tube is thoroughly evacuated and then melt-sealed to a short total length. A vigorous reaction sets in on heating. The entire tube is heated for a while at temperatures up to 300°C in order to achieve product uniformity. The excess  $I_2$ is then allowed to distill off into a somewhat cooler zone, and the tube is quickly cooled and opened.

PROPERTIES:

Brown-black, crystalline powder. Very hygroscopic. Readily soluble in water giving a brown solution which turns gradually green in air. Also soluble in absolute ethanol; insoluble in benzene,  $CCl_4$ ,  $CS_2$ . d 4.2.

REFERENCE:

A. Morette. Comptes Rendus Hebd. Séances Acad. Sci. <u>207</u>, 1218 (1938).

# Vanadium Oxychloride

VOCI

$$V_2O_3 + VCl_3 = 3 VOCl$$
  
149.9 157.3 307.2

A quartz tube, about 180 mm. long and 15 mm. in diameter, is charged with 1 g. of  $V_2O_3$  and 2 g. of VCl<sub>3</sub> under anhydrous

conditions. The tube is thoroughly evacuated, melt-sealed and placed horizontally in a furnace providing a temperature gradient such that the raw material is kept at 720°C and the empty half of the tube at 620°C (see Part I, p. 76 f. and preparation of TiOC1, p. 1209). After 1-2 days, VOC1 forms as a crystalline deposit in the center of the tube and a dense mass on the 720°C side. In addition, small amounts of VC1<sub>2</sub>, VC1<sub>3</sub>, VC1<sub>4</sub> and VOC1<sub>3</sub> are also present. The tube is opened in dry N<sub>2</sub>, and the VC1<sub>4</sub> and VOC1<sub>3</sub> are vaporized in vacuum. The tube contents are then slurried in dimethylformamide, and the VC1<sub>2</sub> and VC1<sub>3</sub> are removed from the VOC1 by repeated slurrying and decantation with this solvent. The VOC1, obtained as the residue, is washed with ethanol and ethyl ether and vacuum-dried.

Alternate methods: Vanadium oxychloride can also be obtained a) by heating VCl<sub>3</sub> and CO<sub>2</sub> [O. Ruff and H. Lickfett, Ber. dtsch. chem. Ges. <u>44</u>, 506 (1911); E. Wedekind and C. Horst, Ber. dtsch. chem. Ges. <u>45</u>, 262 (1912)];

b) by heating VCl<sub>3</sub> in an O<sub>2</sub>-containing N<sub>2</sub> stream;

c) by heating VOCl<sub>2</sub> in a pure N<sub>2</sub> stream [P. Ehrlich and H. J. Seifert, Z. anorg allg. Chem. <u>301</u>, 282 (1959)].

PROPERTIES:

Formula weight 102.41. Brown crystals, the particle size depending on method of preparation. Decomposes at about 600°C into VCl<sub>3</sub> and the oxide; not attacked by  $H_2O$ , hydrochloric acid or alkalies; dissolves in warm conc. HNO<sub>3</sub> and conc.  $H_2SO_4$ . d 3.44. Rhombic crystals, isotypical with FeOCl.

REFERENCES:

H. Schäfer and F. Wartenpfuhl. J. Less-Common Metals <u>3</u>, 29 (1961); P. Ehrlich and H. J. Seifert. Z. anorg. allg. Chem. <u>301</u>, 282 (1959).

# Vanadium Oxydichloride

### VOCl<sub>2</sub>

$$V_2O_5 + 3 VCl_3 + VOCl_3 = 6 VOCl_2$$
  
181.9 472.0 173.3 827.2

A thoroughly ground mixture of 3.6 g. of dry  $V_2O_5$  and 9.4 g. of VCl<sub>3</sub> is placed at the closed end of a 1-m.-longtube, and 0.9 ml. of VOCl<sub>3</sub> is then added. The upper part of the tube must be free of traces of these substances. The tube, filled with air, is melt-sealed, and is covered along its entire length with a sheet-metal jacket; its

lower third, in a slightly inclined position, is then heated to about 600°C with a tubular electric furnace. The sheet-metal jacket provides a temperature gradient along which the product  $VOCl_2$  sublimes out of the hot reaction zone. This procedure requires at least 4-5 days. However, the yield can be increased by longer heating time. Green needlelike crystals of  $VOCl_2$  are deposited in the cold part of the tube. The tube is opened at a suitable spot; the product is suspended in petroleum ether, ethyl ether or  $CCl_4$  to dissolve some adhering  $VOCl_3$ , and then suction-filtered on a coarse fritted-glass filter. The relatively coarse filter separates the  $VOCl_2$  crystals from traces of finely divided hydrolysis products. The  $VOCl_2$  is freed of adhering solvent and stored under anhydrous conditions.

Alternate method: The older method of reducing VOCl<sub>3</sub> with Zn powder in a sealed tube is less efficient (Gmelin-Kraut, Handb. d. anorg. Chem. [Handbook of Inorg. Chem.], 7th Ed., Heidelberg, 1908, Vol. III/2, p. 120).

SYNONYM:

Vanadyl dichloride.

PROPERTIES:

Formula weight 137.86. Shiny green crystals; hygroscopic. d 2.88. Solutions of VOCl<sub>2</sub> in aqueous hydrochloric acid are obtained by adding VCl<sub>4</sub> to  $H_2O$  or by heating  $V_2O_5$  with excess conc. HCl and evaporating most of the excess HCl. The evolution of Cl<sub>2</sub> brought about in this manner can be greatly facilitated by addition of weak reducing agents such as ethanol or  $H_2S$ .

REFERENCE:

H. Funk and W. Weiss. Z. anorg. allg. Chem. 295, 327 (1958).

# Vanadium Oxytrichloride

### VOCl<sub>3</sub>

I.

 $V_2O_5 + 3 SOCl_2 = 2 VOCl_3 + 3 SO_2$ 181.9 356.9 346.6

A flask connected to a reflux condenser via a ground joint is charged with 20 g. of  $V_2O_5$  and 24 ml. of  $SOCl_2$  (equivalent quantities) and heated for 6-8 hours on a water bath under rigorously anhydrous conditions. After rearranging the apparatus for forward distillation, the reaction product is distilled directly from the flask. This method yields pure VOCl<sub>3</sub> provided no excess of SOCl<sub>2</sub> is used. II.

$$V_2O_3 + 3 Cl_2 = 2 VOCl_3 + \frac{1}{2}O_2$$
  
149.9 67.2 l. 346.6

Pellets, obtained by compressing a mixture of  $V_2O_3$  and coal powder, are heated at 500-600°C in a  $Cl_2$  stream. The red-brown reaction product, containing VOCl<sub>3</sub> as well as considerable amounts of VCl<sub>4</sub> and Cl<sub>2</sub>, is repeatedly redistilled over Na metal until it is yellow. The product should not be distilled to dryness because it is likely to ignite. Rigorously anhydrous conditions are required.

Alternate method: The VCl<sub>3</sub> is heated in a stream of O<sub>2</sub>, using the apparatus described for the preparation of VCl<sub>4</sub> from VCl<sub>3</sub> and Cl<sub>2</sub>, p. 1259 [H. Funk and W. Weiss, Z. anorg. allg. Chem. <u>295</u>, 327 (1958)].

SYNONYM:

Vanadyl trichloride.

PROPERTIES:

Formula weight 173.32. Light-yellow liquid. M.p.  $-79.5^{\circ}$ C, b.p. 127°C; vapor pressure (0°C) 4.4 mm., (80°C) 175 mm.; d (0°C) 1.85, (32°C) 1.81. Instantly hydrolyzed by H<sub>2</sub>O; quickly attacked even by atmospheric moisture; violent reaction with Na above 180°C. Soluble in ethanol, ethyl ether and glacial acetic acid.

**REFERENCES:** 

- I. H. Hecht, G. Jander and H. Schlapmann. Z. anorg. Chem. 254, 255 (1947).
- II. W. Prandtl and B. Bleyer. Z. anorg. Chem. <u>65</u>, 153 (1909);
  L. Vanino. Handb. d. präp. Chemie, Anorg. Teil [Handbook of Preparative Chemistry, Inorganic Section], 3rd Ed., Stuttgart, 1925, p. 675; F. E. Brown and F. A. Griffits in: H. S. Booth, Inorg. Syntheses, Vol. I, New York-London, 1939, p. 106 and J. C. Bailar, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 80; A. Morette. Comptes Rendus Hebd. Séances Acad. Sci. <u>202</u>, 1846 (1936).

### Vanadium Dioxychloride

VO<sub>2</sub>Cl

$$\begin{array}{c} \text{VOCl}_{3} + \text{Cl}_{2}\text{O} = \text{VO}_{2}\text{Cl} + 2 \text{ Cl}_{2} \\ 173.3 & 86.9 & 118.4 \end{array}$$

A 250-ml., two-neck flask equipped with a large-diameter inlet tube, with all joints and stopcocks made gas-tight with Hostaflon (Farbwerke Hoechst) or Teflon grease, is purged with dry  $N_2$ . Then pure VOCl<sub>3</sub> (100 ml.) is distilled in, and Cl<sub>2</sub>O gas diluted with O<sub>2</sub> is introduced at room temperature. The Cl<sub>2</sub>O gas mixture is obtained by passing O<sub>2</sub> and Cl<sub>2</sub>(both dried over P<sub>2</sub>O<sub>5</sub>) over HgO. It is advantageous to keep the HgO in a rotatable glass tube surrounded by a cooling jacket (Liebig condenser).

After a while an orange-colored, microcrystalline mass is formed, while the temperature of the reaction mixture increases slightly. The  $O_2$ -Cl<sub>2</sub>O feed is continued until the quantity of the crystals formed makes further feeding impossible. The material is filtered under rigorously anhydrous conditions and then vacuumdried. The product must not be washed with CCl<sub>4</sub>, becaused this solvent slowly reacts with VO<sub>2</sub>Cl even at room temperature, affording phosgene.

The yield, based on  $Cl_2O$ , is nearly quantitative; based on  $VOCl_3$ , it is not higher than 60%.

PROPERTIES:

Orange-red, microcrystalline, very hygroscopic powder. At 150°C, disproportionates into  $V_2O_5$  and  $VOCl_3$ . Sparingly soluble in nonpolar solvents, moderately soluble in ethyl ether, soluble in H<sub>2</sub>O with decomposition. d 2.29.

REFERENCE:

K. Dehnicke. Personal communications, 1960.

# Lower Vanadium Oxides

V6O13

a) A stoichiometric mixture of  $V_2O_5$  and  $V_2O_3$  is heated for 48 hours at not less than 600°C, preferably at 750-800°C, in an evacuated, sealed quartz tube.

b) Or, the  $V_2O_5$  surface is reduced in an SO<sub>2</sub> stream at a temperature somewhat higher than 700°C, and the unreacted  $V_2O_5$  is extracted from the reaction product with conc. ammonia.

PROPERTIES:

Blue-black, crystalline powder. Readily soluble in conc. HNO<sub>3</sub>, sparingly soluble in conc. ammonia and 2 N NaOH. The independent

1266

phase with a monoclinic crystal structure is stable only below about 700°C.

VO<sub>2</sub>

Since the direct reduction of  $V_2O_5$  does not give a well-defined product, this compound is best prepared by synthesis from  $V_2O_5$  and  $V_2O_3$ , suggested a long time ago by Berzelius:

 $\begin{array}{rrrr} V_2O_5 \ + \ V_2O_3 \ = \ 4 \ VO_2 \\ 181.9 & 149.9 & 331.8 \end{array}$ 

An intimate mixture of  $V_2O_5$  and  $V_2O_3$ , in the exact proportion required by the equation, is heated for 40-60 hours at 750-800°C in a small evacuated, sealed quartz tube.

Alternate methods: The  $V_2O_5$  is fused with an excess of crystalline oxalic acid until a greenish-blue, completely water-soluble mass of vanadyl oxalate is obtained. This mass is then calcined to complete decomposition in the absence of air. The VO<sub>2</sub> is obtained as the residue.

PROPERTIES:

Blue-black powder. M.p. 1650°C; d 4.34. Deformed C 4 (rutile) type crystal structure.

 $V_nO_{2n-1}$ 

According to G. Anderson, several lower vanadium oxides have very similar compositions which are intermediate between those of VO<sub>2</sub> and of V<sub>2</sub>O<sub>3</sub> and correspond to the formula  $V_nO_{2n-1}$  (where n = 3, 4, 5, 6, 7 or 8). These oxides have very narrow regions of homogeneity and are obtained by vacuum heating corresponding mixtures of V<sub>2</sub>O<sub>5</sub>, V<sub>2</sub>O<sub>3</sub> and V for 2-20 days at 650-1000°C.

V<sub>2</sub>O<sub>3</sub>

$$V_2O_5 + 2H_2 = V_2O_3 + 2H_2O_181.9 + 44.81. + 149.9 + 36.0$$

The  $V_2O_5$  is reduced in a stream of very pure  $H_2$  in two steps: first, for 2 hours at 600°C (the 658°C m.p. of  $V_2O_5$  should not be exceeded), then for 6 additional hours at 900-1000°C.

For information on the formation of  $V_2O_3$  single crystals, see H. Hahn and C. de Laurent, Angew. Chem. <u>68</u>, 523 (1956).

PROPERTIES:

Dull, black powder. M.p. 1970°C; d 4.87. Crystal structure: alumina type.

vo

$$V_2O_3 + V = 3 VO$$
  
149.9 51.0 200.9

Synthesized from  $V_2O_3$  and V metal powder under vacuum or Ar. The reactants are kept in  $Al_2O_3$  crucibles which, in turn, are inserted into small, evacuated quartz tubes. Or. the apparatus described by Ehrlich for the preparation of TiO (p. 1214) may be used. The optimum reaction temperatures lie between 1200 and 1600°C. A product of greater uniformity is obtained by occasionally interrupting the heating and repulverizing the material. A reaction time of the order of 24 hours at 1200-1300°C is needed, whereas 1 hour is sufficient at 1600°C.

Alternate method: Electrolysis of phosphate melts containing dissolved  $V_2O_5$  [H. Hartmann and W. Mässing, Z. anorg. allg. Chem. <u>266</u>, 98 (1951)].

PROPERTIES:

Gray powder. The region of homogeneity is  $VO_{0.75}-VO_{1.20}$ . Crystal structure: B 1 (NaCl) type.

REFERENCES:

H. Fendius. Thesis, Univ. of Hannover, 1930; W. Klemm and L. Grimm. Z. anorg. allg. Chem. <u>226</u>, 359 (1936); W. Klemm and E. Hoschek. Z. anorg. allg. Chem. <u>242</u>, 63 (1939); W. Klemm and L. Grimm. Z. anorg. allg. Chem. <u>250</u>, 42 (1942); W. Klemm and P. Pirscher. Optik <u>3</u>, 75 (1948); F. Aebi. Helv. Chim. Acta <u>31</u>, 8 (1948); S. S. Todd and K. R. Bonnickson. J. Amer. Chem. Soc. <u>73</u>, 3894 (1951); M. Frandsen. J. Amer. Chem. Soc. <u>74</u>, 5046 (1952); N. Schönberg. Acta Chem. Scand. <u>8</u>, 221 (1954); G. Anderson. Acta Chem. Scand. 8, 1509 (1954).

# Vanadium (III) Hydroxide

### V(OH)<sub>3</sub>

A neutral Na $_3$ VO  $_4$  solution is reduced electrolytically at a mercury cathode.

1268

### A) PREPARATION OF THE VANADATE SOLUTION

 $V_{2}O_{5} + 6 NaOH = 2 Na_{3}VO_{4} + 3 H_{2}O$ 181.9 240.0 367.9 54.0

Concentrated NaOH is prepared from equal weights of NaOH and  $H_2O$ , and the solution is filtered in the absence of  $CO_2$  to remove the Na<sub>2</sub>CO<sub>3</sub> contaminant. The strength of the solution is then

Fig. 302. Preparation of vanadium (III) hydroxide. A) carbon anode; D) clay diaphragm; E) electrolyte solution; G) rubber stopper; N) standard cell; R) N<sub>2</sub> inlet type; S) electrolyte bridge. determined by analysis. The amount of  $V_2O_5$  required by the above equation is then dissolved in this solution, and 1 M Na<sub>3</sub>VO<sub>4</sub> is prepared by dilution with water.

#### B) ELECTROLYSIS

An electrolysis cell of the type shown in Fig. 302 is filled with very pure Hg and the Na<sub>3</sub>VO<sub>4</sub> solution. The cell is closed off with rubber stopper G, and the anode space within the clay diaphragm D is filled with 1 M Na<sub>2</sub>SO<sub>4</sub> solution. Very pure  $N_2$ is bubbled (via inlet tube R) through the electrolyte solution for the first 15 minutes, and is passed over the liquid surface for the remainder of the electrolysis. Electrolysis is carried out with a current of 0.1 amp. at a cathode potential of -1.70 v. (checked by means of the standard cell N). Brown-green V(OH)<sub>3</sub> precipitates, and the electrolyte assumes a brownred color. The electrolysis is discontinued after 5-10 hours, and V(OH)<sub>3</sub> isolated in

the absence of air. This is accomplished by first draining the Hg through the stopcock, then suction-filtering the  $V(OH)_3$  suspension under a N<sub>2</sub> atmosphere and thoroughly washing the precipitate with deaerated H<sub>2</sub>O. Finally the  $V(OH)_3$  is dried in high vacuum.

#### PROPERTIES:

Formula weight 101.97. Brass-colored, crystalline; readily oxidized by  $O_2$ .

#### REFERENCE:

N. Konopik and A. Neckel. Monatsh. Chem. 88, 917 (1957).



# Vanadium (V) Oxide

### $V_2O_5$

Highest grade  $V_2O_5$  is obtained by calcining NH<sub>4</sub>VO<sub>3</sub> at 500-550°C. Traces of N usually contaminating the product thus obtained can be virtually eliminated by heating for 18 hours at 530-570°C in a moist O<sub>2</sub> stream.

PROPERTIES:

Formula weight 181.9. Red to orange-yellow powder. M.p.  $674^{\circ}$ C. Slightly soluble in water: 0.07 g./100 g. H<sub>2</sub>O; d 3.36. Readily soluble in alkali hydroxide solutions, acids and ethanol. Crystal structure: orthorhombic.

Supported  $V_2O_5$  ( $V_2O_5$  catalysts)

I. ON ASBESTOS

A solution of 5g. of NH<sub>4</sub>VO<sub>3</sub> in 100 ml. of boiling H<sub>2</sub>O is reduced with NH<sub>4</sub>HSO<sub>3</sub> and treated with sulfuric acid until the solution turns a pure blue-violet. Asbestos (20 g.) is added; the mixture is allowed to boil for 10-15 minutes and is then cooled to 40-50 °C; it is then rendered strongly alkaline by addition of conc. NH<sub>3</sub>, whereby the V(OH)<sub>3</sub> precipitates onto the asbestos. The latter is dried and again treated with unreduced NH<sub>4</sub>VO<sub>3</sub>, whereupon the asbestos and the flask wall become violet-blue. The asbestos mass is then pulled apart into small clumps, dried and calcined at 500-600°C; its V<sub>2</sub>O<sub>5</sub> content may be as high as 50%.

II. ON CERAMIC MATERIALS

To produce a homogeneous deposit of  $V_2O_5$  on ceramic materials (e.g., firebrick), 2-3 equivalents of a mineral acid is added to a NH<sub>4</sub>VO<sub>3</sub> solution. The resulting dark-yellow solution, which contains colloidal  $V_2O_5 \cdot aq$ ., is boiled in the presence of the ceramic material (heating on the water bath is insufficient), thereby causing precipitation onto the carrier of a yellow-red, strongly adhering layer of  $V_2O_5$ . Glass (not porcelain) reaction vessels should be used in this process.

**REFERENCES**:

N. Yefremov and A. Rozenberg. Khim. Prom. <u>4</u>, 129; Chem. Zentr. <u>1927</u>, II, 1994; I. Adadurov and G. Boreskov. Khim. Prom. <u>6</u>, 732; Chem. Zentr. <u>1929</u>, II, 2926. Colloidal V205

### I. METHOD OF BILTZ

a) A mortar is used to grind 1 g. of  $NH_4VO_3$  with some  $H_2O$ , and 10 ml. of 2 N HCl is added while stirring with the pestle. The red precipitate formed and the supernatant are transferred onto a filter, the liquid is filtered off, and the solid is washed with  $H_2O$ . After a while, the initially clear filtrate becomes reddish and turbid. The precipitate is then transferred from the filter to an Erlenmeyer flask (using water from a wash bottle), and the volume is adjusted to 100 ml. The quantities of material used may be increased up to 200-fold, provided the proportions are kept the same. After a few hours the conversion of the precipitate to a clear orange-red  $V_2O_5$  sol is complete. Fibrillar birefringence becomes evident only after prolonged standing.

b) In the modification of Humphry, 0.5 g. of  $NH_4VO_3$  and 2 ml. of nitric acid (1 vol. of conc.  $HNO_3 + 10$  vol. of  $H_2O$ ) are ground together and then an additional 2 ml. of  $HNO_3$  is added. The mixture is filtered, and the  $V_2O_5$  formed is washed until it starts passing through the filter; it is then shaken with 200 ml. of  $H_2O$  and allowed to stand for 14 days.

PROPERTIES:

Because of the fibrillar structure of the particles, the sol exhibits strong streaming birefringence and on aging shows an increase in fibril length. Part of the  $V_2O_5$  is always molecularly dissolved, in an amount increasing symbatically with the total concentration. The sol absorbs electrolytes strongly, and always contains some vanadium (IV).

II. METHOD OF PRANDTL AND HESS: SAPONIFICATION OF TERT-BUTYL ORTHOVANADATE

 $2 (C_4H_9)_3 VO_4 + 3 H_2O = V_2O_5 + 6 C_4H_9OH$ 572.6 54.0 181.9 444.7

a) tert-Butyl alcohol and  $V_2O_5$  are refluxed for several hours, affording a light-yellow solution of tert-butyl orthovanadate containing 5.3 g. of  $V_2O_5$  per 100 g. The solution is filtered and subjected to fractional vacuum distillation in an Anshütz flask. This ester is sometimes repurified by a second vacuum distillation. Ester properties: b.p. 117°C (15 mm.), 132°C (32 mm.); m.p. 45-47°C.

b)  $H_2O$  is added to the ester; this produces a very loose (swollen) orange-colored precipitate. On boiling with a large quantity of  $H_2O$ , a clear, colloidal  $V_2O_5$  solution is obtained. The alcohol (b.p. 82°C) is completely eliminated by the boiling.

PROPERTIES:

Electrolyte-free, virtually monodisperse sol; does not age appreciably.

Alternate methods: a) Pouring molten  $V_2O_5$  into  $H_2O$  gives a fairly polydisperse sol, which is electrolyte-free and undergoes little or no aging [E. Müller, Kolloid-Z. 8, 302 (1911)].

b) Dissolving  $V_2O_5$  in dilute  $H_2O_2$  and boiling the solution affords a strongly polydisperse sol, which is electrolyte-free and does not age appreciably [W. Ostermann, Jahrb. d. philos. Fak. Göttingen II, 265 (1921)].

REFERENCES:

General: H. Gessner. Kolloid-Beih. 19, 213 (1924).

- I. W. Biltz, Ber. dtsch. chem. Ges. <u>37</u>, 1095 (1904); E. Sauer. Kolloidchem. Praktikum [Lab. Manual of Colloid Chemistry], Berlin, 1935; R. H. Humphry. Proc. Phys. Soc. (London) <u>35</u>, 217 (1923).
- II. W. Prandtl and L. Hess. Z. anorg. allg. Chem. 82, 102 (1913).

# Ammonium Metavanadate

### NH<sub>4</sub>VO<sub>3</sub>

Ammonium metavanadate is a common commercial product, but its purity usually leaves something to be desired. Since it is easily prepared from  $V_2O_5$  (see method II) and in turn readily yields  $V_2O_5$  on calcination, it plays an important role in the preparation and purification of vanadium compounds. Only the purification of NH<sub>4</sub>VO<sub>3</sub> is treated here.

I. A saturated  $NH_4VO_3$  solution is prepared in boiling, weakly ammoniacal  $H_2O$ . About 500 ml. of  $H_2O$  is required for 25 g. of  $NH_4VO_3$  but, to avoid excessive hydrolysis, it is better to use more water than to prolong the boiling. The solution is filtered hot through a fine-pore fritted glass filter, 10% of its weight of solid  $NH_4Cl$  is added to the filtrate, and the mixture is cooled to  $0^{\circ}C$ . The precipitating  $NH_4VO_3$  is allowed to stand at  $0^{\circ}C$  (1 hour); it is then suction-filtered and washed with a small amount of icecold, ammoniacal  $H_2O$  and finally with some ice-cold pure water.

If necessary, this recrystallization may be repeated several times. Complete removal of alkali ions and traces of  $V_2O_5 \cdot xH_2O$  (formed on slight hydrolysis) is extremely difficult.

II. If a smaller volume of liquid is desired,  $V_2O_5$  is dissolved in Na<sub>2</sub>CO<sub>3</sub> as NaVO<sub>3</sub>, and NH<sub>4</sub>VO<sub>3</sub> is precipitated from this solution by addition of NH<sub>4</sub>Cl.

Thus 25 g. of  $V_2O_5$  is added, in small portions and with stirring, to a boiling solution of 17.5 g. of anhydrous Na<sub>2</sub>CO<sub>3</sub> in 125 ml. of H<sub>2</sub>O. After the CO<sub>2</sub> evolution has subsided, saturated KMnO<sub>4</sub>

solution is added to the reaction mixture in an amount just sufficient to discharge the green-blue color stemming from partial reduction of the vanadium. Undissolved  $V_2O_5$  and  $MnO_2$  are carefully suctionfiltered on a fine-pore fritted glass filter until the filtrate is completely clear. The residue is washed with  $H_2O$  until  $H_2O_2$ no longer gives a positive reaction for vanadium. The filtrate, which amounts to about 125-150 ml., is heated to 60°C and then poured all at once into a hot solution of 75 g. of  $NH_4Cl$  in 125 ml. of  $H_2O$ . Precipitation of  $NH_4VO_3$  starts immediately and is complete after a few hours. Some of the salt adheres fairly tenaciously to the glass walls. The salt is suction-filtered and washed with small portions of  $H_2O$  until the washings are free of chloride ion. It is then dried in air at a temperature below 40°C. An almost purewhite salt, still containing some Na (about 0.3% NaCl) is obtained in 80% yield.

The NH<sub>4</sub>VO<sub>3</sub> can be recrystallized under similar conditions. For example, 25 g. of the salt is dissolved in a solution of 16 g. of Na<sub>2</sub>CO<sub>3</sub> in 125 ml. of H<sub>2</sub>O with mild heating (to 30-40 °C). The mixture is then carefully filtered and precipitated at once with NH<sub>4</sub>Cl as described above.

PROPERTIES:

White, crystalline salt, often yellowish due to slight traces of  $V_2O_5$ . Solubility: (15°C) 0.52 g., (32°C) 1.0 g., (50°C) 1.6 g. per 100 g. of  $H_2O$ . d 2.33. Liberates NH<sub>3</sub> above 50°C. Readily converted to  $V_2O_5$  when heated dry.

REFERENCES:

- I. L. Vanino. Handb. d. präp. Chem., Anorg. Teil [Handbook of Preparative Chemistry, Inorg. Part], 3rd Ed., Stuttgart, 1925, p. 672; M. Lachartre. Bull. Soc. Chim. France [4] <u>35</u>, 321 (1924).
- II. R. H. Baker, H. Zimmermann and R. N. Maxson in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 117; experimental data and personal communications from several laboratories.

# Alkali Vanadates

Among the alkali vanadates, only NaVO<sub>3</sub> is commercially available. However, the number of existing defined alkali vanadates is rather large. Only the system  $K_2O-V_2O_5$  has been subjected to a systematic preparative study. It may be assumed that the other systems have similar structures.

A stoichiometric mixture of  $K_2CO_3$  and  $V_2O_5$  (for example, a total of 5-6 g.) is heated in an open crucible. The crucible material

is Pt or, if the mixtures are rich in alkali, an 80% Au-20% Pd alloy. The temperature should not be increased at a rate faster than  $10^{\circ}$ C/min., in order to prevent a too vigorous reaction which would result in loss by spattering. At low alkali contents, the maximum required temperature is about 500°C, while 1000°C is needed when the alkali content is high. Several hours (8-24) of heating are required at these temperatures. The product is cooled to below 350°C in a desiccator to prevent moisture absorption. The preparation must be modified somewhat, depending on raw material composition (see the original literature).

The phase diagram of the  $K_2O-V_2O_5$  system shows the existence of the following potassium vanadates:  $K_2O \cdot 4 V_2O_5$  (m.p. 520°C, incongruent);  $K_2O \cdot V_2O_5$  (m.p. 520°C); 16  $K_2O \cdot 9 V_2O_5$  (m.p. 696°C, incongruent); 2  $K_2O \cdot V_2O_5$  (m.p. 910°C); 3  $K_2O \cdot V_2O_5$ (m.p. ~1300°C).

REFERENCE:

F. Holtzberg, A. Reisman, M. Berry and M. Berkenblit. J. Amer. Chem. Soc. <u>78</u>, 1536 (1956).

# Vanadium Sulfides

All vanadium sulfides can be synthesized from the elements. An intimate mixture of the finely pulverized components, in the proper proportions, is placed in sintered clay tubular crucibles; these are inserted in quartz tubes, which are then evacuated and melt-sealed. The tubes are then slowly heated and finally maintained for a long time at a maximum temperature of 1000-1300°C. Contact between the vanadium metal and the quartz must definitely be avoided.

This procedure yields  $V_3S$ , VS and  $V_2S_3$ .

 $3V + S = V_3S$ ; V + S = VS;  $2V + 3S = V_2S_3$ 152.9 32.1 185.0 60.0 32.1 92.1 101.9 96.2 198.1

All the sulfides, as phases of the V-S system, have more or less wide regions of homogeneity.

Other preparative methods can also be used to obtain particular sulfide phases.

VS

$$V_2S_3 = 2VS + S_{198,1} + 166.0 + 32.1$$

Thermal decomposition of  $V_2S_3$  at 1000°C in a H<sub>2</sub> stream yields a product of composition  $VS_{1,02}$  in 20 hours.

1274

Pure VS can also be prepared by prolonged calcination of  $V_2O_5$  in a stream of  $H_2S$ .

**PROPERTIES:** 

VS exists, as a phase of the V-S system, between VS<sub>1.00</sub> and VS<sub>1.16</sub>. d 4.51. It has a B8 (NiAs) type crystal structure with voids.

 $V_2S_3$ 

$$V_2O_3 + 3 H_2S = V_2S_3 + 3 H_2O_{149.9}$$
  
102.2 198.1 54.0

A thin layer of about 0.5 g. of  $V_2O_3$  is spread in a porcelain boat, which is then inserted into a suitable tube and heated for 10 hours at 750°C in a moderately fast H<sub>2</sub>S stream, predried over silica gel. At the end of this heating period, the tube is cooled in an H<sub>2</sub>S stream and the  $V_2S_3$  removed from the boat. A uniform and thin layer of  $V_2O_3$  is essential to achieve a reasonable reaction time. Under the same conditions, 2-3 g. of  $V_2O_3$  requires 2 days for complete conversion to the sulfide. Increasing the temperature to 850°C reduces the reaction time to a few hours, but the end product contains somewhat less S than required by the formula  $V_2S_3$  (e.g.,  $VS_{1.47}$ ).

In addition,  $V_2O_5$  can also be used as the starting material, by heating it at 700°C in a stream of  $CS_2$  vapor. For data concerning the formation of  $V_2S_3$  single crystals, see H. Hahn and C. de Laurent, Angew. Chem. <u>68</u>, 523 (1956).

PROPERTIES:

Homogeneous between VS<sub>1.17</sub> and V<sub>1.53</sub>. Dark-gray powder. Quite resistant to dil. acids; in contrast to VS<sub>4</sub>, insoluble in dilute sodium hydroxide. d 3.7.

 $VS_4$ 

A mixture of  $V_2S_3$  and excess S, corresponding to the approximate formula  $VS_{20}$ , is heated for 15 hours at 400°C in a sealed tube, followed by a 12-hour annealing period at 90°C (to convert the excess S into the soluble  $\alpha$ -form). The reaction product is then exhaustively extracted in a Soxhlet apparatus, with the  $VS_4$ remaining as the residue. Extending the heating period (to several months) affords larger crystals. PROPERTIES:

Formula weight 179.19. Black powder. Composition sometimes does not correspond exactly to the formula. Unstable above 500°C, decomposing into  $V_2S_3$  and S. Quite resistant to dilute acids; readily and completely soluble in sodium hydroxide, yielding a red solution. VS<sub>3.65</sub>: d 2.8.

REFERENCES (all sulfides):

W. Biltz and A. Köcher. Z. anorg. allg. Chem. <u>241</u>, 324 (1939); E. Hoschek and W. Klemm. Z. anorg. allg. Chem. <u>242</u>, 49 (1939);
W. Klemm and E. Hoschek. Z. anorg. allg. Chem. <u>226</u>, 362 (1936); B. Pedersen and F. Grønvold. Acta Crystallogr. <u>12</u>, 1022 (1959); B. Pedersen. Acta Chem. Scand. <u>13</u>, 1050 (1959);
G. M. Loginov. Zh. Neorg. Khimii <u>5</u>, 221 (1960).

# Vanadium Selenides

I.

 $V_2O_3 + 3H_2 + 3Se = V_2Se_3 + 3H_2O$ 149.9 67.2 L 236.9 338.8 54.0

About 0.5 g. of  $V_2O_3$ , which during its preparation has been heated not higher than 500-600°C so that is an active product capable of fast reaction, is placed in a small porcelain boat, which is inserted into a quartz tube. A larger boat containing Se is placed ahead of the one containing  $V_2O_3$ , and a stream of very pure  $H_2$ is passed through the tube. The  $V_2O_3$  zone is first heated to 600°C, and then gradually to 900°C, using a small tubular electric furnace; simultaneously, the Se is vaporized by heating with a gas burner. The section of the quartz tube extending beyond the furnace is cooled with a cooling coil. After passing over the reaction product, most of the Se condenses in this section. After the reaction, the tube is allowed to cool and the product is repulverized; the selenation is repeated twice.

The composition of the products thus obtained varies markedly: when prepared at 800°C, the end product is  $\sim VSe_{1.9}$ , at 1000°C  $\sim VSe_{1.4}$ .

II. Heating the above products with a suitable excess of Se in a short sealed quartz tube at a temperature which, depending on the Se content, should be between 600 and  $800^{\circ}$ C affords products with a higher Se content.

III. Thermal degradation of the products of the first selenation, by heating in a high vacuum at 1000-1100°C for several days, affords  $VSe_{1.0}$ .

**PROPERTIES:** 

The V-Se system has 3 stable phases with very broad homogeneity regions.

 $\alpha$ -Phase (VSe), from VSe<sub>1.0</sub> to VSe<sub>1.25</sub>. Dull-gray powder. d 5.94. B8 (NiAs) type crystal structure, with voids.

 $\beta$ -Phase (V<sub>2</sub>Se<sub>3</sub>), from VSe<sub>1.25</sub> to VSe<sub>1.60</sub>. Gray powder with metallic luster. d 5.87.

 $\gamma$ -Phase (VSe<sub>2</sub>), from VSe<sub>1.62</sub> to VSe<sub>1.97</sub>. Gray, small, leaflike crystals with metallic luster. d 5.79. Crystal structure: C6 (CdI<sub>2</sub>) type.

**REFERENCE:** 

E. Hoschek and W. Klemm. Z. anorg. allg. Chem. 242, 49 (1939).

### Vanadium (II) Sulfate

### VSO<sub>4</sub> · 6 H<sub>2</sub>O

Produced by electrolytic reduction of VOSO<sub>4</sub>, followed by ethanol precipitation of VOSO<sub>4</sub>  $\cdot$  6 H<sub>2</sub>O from the resulting vanadium (II) solution.

 $\begin{array}{c} VOSO_4 \cdot 3 \text{ H}_2O \rightarrow VSO_4 \cdot 6 \text{ H}_2O \\ 217.1 & 255.1 \end{array}$ 

A) ELECTROLYTIC REDUCTION

The electrolysis vessel consists of a glass cylinder, 5 cm. in diameter and 10 cm. high, such as, for example, a small Pyrex pressure vessel. This vessel is closed with a five-hole rubber stopper to accommodate the cathode stem, thermometer, diaphragm, and  $N_2$  inlet and outlet tubes. A suitable outlet tube is a small fermentation tube, which serves as protection against air and should, if possible, be drawn out into an outward-pointing capillary.

The cathode is made by bending a lead strip  $(3 \times 25 \times 80 \text{ mm.})$ into a cylinder so that it encloses the diaphragm. A hole is then drilled near one end of the Pb strip, and a lead rod (serving as a bus bar) is attached to it by hammering in place. The anode also consists of a lead rod, which can be prepared by pouring molten Pb into a glass tube, while cautiously fanning the latter with a flame, and then cooling and breaking the tube. Prior to electrolytic reduction, the cathode is pretreated, according to the method of Tafel, by using the cathode as an anode in 2N H<sub>2</sub>SO<sub>4</sub> until it is uniformly coated with brown PbO<sub>2</sub>. The diaphragm consists of a glass tube, about 16 cm. long and 2 cm. I.D., with a fine fritted-glass disk sealed to its lower end. A 2M VOSO<sub>4</sub> solution, which is also 2N in  $H_2SO_4$ , is prepared by dissolving 66 g. of VOSO<sub>4</sub> · 3  $H_2O$  (for preparation, see p. 1285) in  $H_2O$ , adding 8.5 ml. (15 g.) of conc.  $H_2SO_4$ , and diluting to 150 ml. with  $H_2O$ .

The diaphragm is filled with 6N  $H_2SO_4$  to the level of the VOSO<sub>4</sub> solution in the cathode space. This is best done by pouring both solutions in at the same time. Electrolysis is started at 0.3 amp. and about 5 v. Cooling is unnecessary, since the temperature does not exceed 30°C. If the current should exceed 0.3 amp. during the first few hours, it must be readjusted to this value. During the intermediate reduction period, the current drops to 0.2 amp., while the voltage increases. The current should, however, not be raised to 0.3 amp. During reduction, the dark-blue VOSO<sub>4</sub> solution first turns dark blue-green, then later dark and opaque. In the final reduction period the current again rises while the voltage drops. Reduction is complete when the solution is red-violet. Reduction time: 55-60 hours. During electrolysis, a slow stream of O<sub>2</sub>-free N<sub>2</sub> or CO<sub>2</sub> is passed through the electrolyte at the cathode.

If only the reduced vanadium (II) solution is needed and precipitation of crystalline  $VSO_4$  is not desired, the electrolyte solution can be protected from atmospheric  $O_2$  in a simpler fashion, by covering it with a xylene layer about 2 cm. deep.

If desired, the electrolysis may be continued at 0.01-0.02 amp. and 3 v. for several months; a completely reduced VSO<sub>4</sub> solution is thus always available for use. To remove solution as needed, the gas outlet tube is pulled out of the rubber stopper and a pipette inserted in its place. Virtually no oxidation takes place if the VSO<sub>4</sub> solution aspirated into the pipette is immediately allowed to run out into another vessel under an inert gas blanket.

A low current density is necessary in order to obtain a relatively concentrated  $VSO_4$  solution with a minimum  $H_2SO_4$  content, as required for the precipitation with ethanol described below.

# B) PRECIPITATION WITH ABSOLUTE ETHANOL

The glass apparatus shown in Fig. 303 is suitable for the isolation of crystalline VSO<sub>4</sub>. Its main section is adapter k, bent at an angle of about 100°. The ends of k are connected via standard taper joints to the other parts of the apparatus. The joints are lightly lubricated with stopcock grease. The entire apparatus is fastened to a cross-shaped supporting rack of iron bars by means of two common clamps (not shown in the drawing), one holding the neck of the round-bottom flask p and the other the receiver m. The rack can be rotated, with moderate resistance, about its axis (which is perpendicular to the plane of the drawing).

The gas inlet tube e is connected via a pressure hose to a twoway stopcock which connects the system either to an oil-type vacuum pump or to a source of pure,  $P_2O_5$ -dried  $N_2$ . The  $N_2$  must be very pure; it is pressurized to 0.2 atm. gauge by means of a Hg or  $H_2O$  leveling device (2-m. water column), and then introduced into the apparatus. A two-liter flask, serving as a  $N_2$ surge vessel, is inserted between the  $N_2$  purifier and the  $P_2O_5$ drying train.



Fig. 303. Preparation of vanadium (II) sulfate. k) bent adapter;  $s_1$ ) 29/26 joint;  $s_2$ ) 24/25 joint;  $s_3$ and  $s_4$ ) 12/18 joints; e) gas inlet tube with stopcock  $h_1$ ; r) 250-ml. round-bottom flask; f) tubular adapter with fritted-glass disk; m) receiver;  $h_1$ and  $h_2$ ) stopcocks.

The apparatus is completely purged of air by alternate evacuation and flushing with nitrogen. Then 20 ml. of 2 M VSO<sub>4</sub> solution is removed from the electrolysis vessel with a pipette, the stopper  $s_3$ is removed, the pipette is immediately inserted through this opening until it almost touches the bottom of flask r, and the solution is allowed to flow out. With stopcock  $h_1$  open, the filling operation is carried out in a countercurrent nitrogen stream.

In the same manner 40 ml. of absolute ethanol, previously deaerated by boiling while passing through pure dry nitrogen, is added to flask r. Stopper  $s_3$  is immediately closed, and the entire apparatus vigorously shaken for 5 minutes by a back-and-forth movement of the cross arm. Solid, granular VSO<sub>4</sub> · 6 H<sub>2</sub>O begins to precipitate within a few seconds. Stopcock  $h_2$  is now connected

via a rubber hose to a wash bottle containing some water and also serving as a bubble counter and liquid seal. With stopcocks  $h_1$  and  $h_2$  open, the pale-purple mother liquor is decanted by carefully tilting the cross arm. The salt precipitate in r is washed by vigorous shaking with the liquid quantities indicated below and decanting the used liquid before each new addition. The wash liquid is introduced, as indicated above in the case of the VSO<sub>4</sub> solution, in a countercurrent  $N_2$  stream by means of a pipette inserted through s3. The wash liquids are introduced in the following order:  $2 \times 15$  ml. of deaerated absolute ethanol; 25 ml. of the same ethanol plus 10 ml. of ab solute ether; 10 ml. of ethanol plus 25 ml. of ether; 5 ml. of ethanol plus 15 ml. of ether; 3 ml. of ethanol plus 25 ml. of ether. Finally the salt is transferred onto the fritted-glass filter f with an additional 25 ml. of absolute ethyl ether. The ether adhering to the substance is removed by continuing the  $N_2$  purge stream (about one hour). Flask *m* is replaced with the  $N_2$ -filled drying vessel t shown in Fig. 304 while continuing the N<sub>2</sub> stream via  $h_3$ . Vessel t is charged to one third of its capacity with a  $P_2O_5$ -pumice drying mixture, and any oxygen present in the latter is removed by alternate evacuation and purging with  $N_2$ . Then the drying vessel and the tube f are detached from adapter k at joint  $s_2$ , f is immediately closed off with a ground stopper, and the system is connected to the oil vacuum pump by way of  $h_3$  (resetting the two-way stopcock). After evacuation, stopcock ha is closed and the drying vessel with the product is allowed to stand for 5 days at 25 °C (drying can also be accomplished, without the P<sub>2</sub>O<sub>5</sub>-pumice mixture, by immersing the evacuated drying vessel in liquid nitrogen). The light red-violet product is stored in container n, shown in Fig. 304, which provides protection from oxygen and enables one to remove the product when desired. To transfer the product to n, the latter is evacuated, then filled with N<sub>2</sub>; the drying vessel is also filled with N<sub>2</sub> through  $h_3$ , the glass stopper is removed from f, and the storage container n (which carries a male joint), from which the two-joint adapter and bent tube p have just been removed, is inserted in its place. During this operation, the  $N_2$ stream is introduced via stopcock  $h_4$  of the storage container, as well as via  $h_3$ . The substance is transferred to n from the fritted-glass tube f by turning the tube upside down. The storage container is then detached from the tube and closed off with the two-joint adapter and bent tube p. To remove any  $O_2$  entrained during the transfer, container n is immediately evacuated and filled with N<sub>2</sub>; these two operations are then repeated twice. Approximately 10 g. of  $VSO_4 \cdot 6 H_2O$  is obtained (90% yield, based on the 20 ml. of 2 M  $VSO_4$  solution used).

Short exposure of the product to air does not affect its storage stability provided the last traces of  $O_2$  are removed by repeated

purging of the system with  $N_2$ . The product is stable for several months, even if frequently sampled.



Fig. 304. Drying tube and storage container for vanadium (II) sulfate. f) fritted-glass tube as in Fig. 303; t) drying tube containing the  $P_2O_5$ -pumice mixture; n) storage tube p) bent tube for filling operations involving small amounts of substance;  $s_2$  and  $s_6$ ) standard taper joints, as in Fig. 303.

To remove some product from the storage container, the estimated amount is transferred to the bent tube p by inclining and tapping n (Fig. 304). Nitrogen is introduced via  $h_4$ , and tube p containing the product is quickly replaced by an identical empty tube. The closed-off storage container is then evacuated twice and filled with  $N_2$ ; at the same time, the open, previously tared bent tube p containing the VSO  $_4 \cdot 6$  H<sub>2</sub>O is quickly weighed (to ascertain the weight of sample in it), and is then quickly connected to the N<sub>2</sub>filled apparatus (Fig. 303) by inserting its male joint into joint  $s_3$ . By quickly turning the cross arm, the salt is poured from the bent tube p into the flask r, whereupon the system (of Fig. 303) is reevacuated via  $h_1$  and refilled with N<sub>2</sub>. The VSO<sub>4</sub> · 6 H<sub>2</sub>O can then again be dissolved in O<sub>2</sub>-free H<sub>2</sub>O, added from a pipette in a countercurrent stream of N<sub>2</sub>. In this manner, an H<sub>2</sub>SO<sub>4</sub>-free solution of  $VSO_4$  is obtained; this can be used to prepare other vanadium (II) compounds in the same apparatus while retaining the certainty that air is completely absent.

Alternate method: A solution of VO in  $H_2SO_4$  is prepared in the absence of air, and is then evaporated in vacuum [C. M. French and J. P. Howard, Trans. Faraday Soc. 52, 712 (1956)].

PROPERTIES:

Light red-violet, fine crystalline powder, oxidized to brown even in dry air. Readily soluble in deaerated  $H_2O$ , yielding a red-violet solution.

A heptahydrate VSO  $_4 \cdot 7~\mathrm{H_2O}$  may also be formed under other conditions.

REFERENCES:

J. Dehnert. Thesis, Univ. of Jena, 1952 S. Herzog. Z. anorg. allg. Chem. <u>294</u>, 155 (1958); L. Malatesta. Gazz. Chim. Ital. <u>71</u>, 615 (1941); J. Meyer and M. Aulich. Z. anorg. allg. Chem. <u>194</u>, 278 (1930); A. Piccini and L. Marino. Z. anorg. Chem. <u>50</u>, 49 (1906).

# Hydrogen Disulfatovanadate (III)

### $HV(SO_4)_2 \cdot 4 H_2O$

A paste obtained by stirring 10 g. of  $V_2O_5$  and 36 g. of conc.  $H_2SO_4$  is heated for a while on a water bath and allowed to stand until the next day. The mixture is then treated with 80 ml. of  $H_2O$  and reduced on the water bath by bubbling through it a stream of  $SO_2$ . Reduction to  $VOSO_4$  is complete in a few minutes. Excess  $SO_2$  is boiled off and the product electrolytically reduced to the trivalent state, using the apparatus described for ammonium vanadium (III) alum (p. 1284). The resulting green solution is filtered on a fine-pore fritted-glass disk and allowed to stand in a vacuum desiccator over  $H_2SO_4$ . After a few days, a green crystalline powder separates out; it is stirred with a large quantity of ethanol, suction-filtered and then thoroughly washed with ethanol. The product is dried over  $H_2SO_4$  in a  $CO_2$ -filled desiccator.

SYNONYM:

Disulfatovanadic (III) acid.

PROPERTIES:

Formula weight 316.14. Green crystalline powder; can be stored for extended periods of time in closed bottles even in the presence of air.

Other compounds: The above procedure yields  $NH_4V(SO_4)_2$ . 4 H<sub>2</sub>O when 12 g, of  $NH_4VO_3$  is used in place of 10 g, of  $V_2O_5$ .

The hexahydrate,  $HV(SO_4)_2 \cdot 6 H_2O$ , is formed if 150 g. of  $H_2SO_4$  is used; the salt  $NH_4V(SO_4)_2 \cdot 6 H_2O$  results when  $NH_4VO_3$  is reacted with the stoichiometric amount of  $H_2SO_4$ .

Sulfates of trivalent vanadium can be obtained using hydrazine as the reducing agent and glacial acetic acid as the reaction medium. An intermediate,  $V(CH_3COO)_3$ , is formed under these conditions. Hydrates of  $V_2(SO_4)_3$ , for example,  $V_2(SO_4)_3 \cdot 9 H_2O$ , can be prepared in this manner.

The anhydrous compounds  $HV(SO_4)_2$  and  $V_2(SO_4)_3$  can be prepared by using conc.  $H_2SO_4$ , or by thermal dehydration. See Meyer and Markowicz.

**REFERENCES**:

J. Meyer and E. Markowicz. Z. anorg. allg. Chem. <u>157</u>, 211 (1926);
 J. T. Brierley. J. Chem. Soc. <u>49</u>, 823 (1886); A. Stähler and
 H. Wirthwein. Ber. dtsch. chem. Ges. <u>38</u>, 3970 (1905); J. Dehnert. Personal communication, 1951.

### Ammonium and Potassium Disulfatovanadate (III)

### $NH_4V(SO_4)_2$ , $KV(SO_4)_2$

 $NH_4V(SO_4)_2$ 

A paste prepared by stirring 12 g. of  $\rm NH_4VO_3$  with some  $\rm H_2O$ is slowly added to 300 ml. of 2 N H<sub>2</sub>SO<sub>4</sub>. The resulting pure yellow solution is mixed with 200 ml. of saturated SO<sub>2</sub> solution and 40 g. of  $(\rm NH_4)_2SO_4$ . The blue solution obtained is evaporated, first on a water bath, then over an open flame, until a blue salt begins to precipitate. Concentrated H<sub>2</sub>SO<sub>4</sub> (30-50 ml.) is added and the heating, during which fumes are evolved, continued for a while. The mixture is allowed to cool overnight and is then taken up in H<sub>2</sub>O. The residue is suction-filtered, triturated with H<sub>2</sub>O, reboiled with H<sub>2</sub>O, thoroughly washed, and then dried over H<sub>2</sub>SO<sub>4</sub> in a vacuum desiccator. Yield: 4.7 g. of NH<sub>4</sub>V(SO<sub>4</sub>)<sub>2</sub>.

 $KV(SO_4)_2$ 

The K salt is prepared similarly, by evaporating a mixture of 200 ml. of 2 N H<sub>2</sub>SO<sub>4</sub>, 10 g. of vanadyl sulfate, 21.1 g. of K<sub>2</sub>SO<sub>4</sub> and 10 ml. of sulfurous acid on a water bath, then adding 10 ml. of conc. H<sub>2</sub>SO<sub>4</sub> and heating for a while, while fumes are evolved. After cooling, 400 ml. of H<sub>2</sub>O is added and the mixture boiled for a short time. The green product is washed with H<sub>2</sub>O, suction-filtered and vacuum-dried over H<sub>2</sub>SO<sub>4</sub>.

#### PROPERTIES:

Green, crystalline powder. Insoluble in  $H_2O$  and acids; attacked and decomposed by alkali.

1283

**REFERENCES**:

A. Sievers and E. L. Müller. Z. anorg. allg. Chem. <u>173</u>, 313 (1928);
A. Rosenheim and H. Y. Mong. Z. anorg. allg. Chem. <u>148</u>, 25 (1925);
V. Auger. Comptes Rendus Hebd. Séances Acad. Sci. <u>173</u>, 306 (1921).

 $NH_4V(SO_4)_2 \cdot 12H_2O$  (Alum)

A mixture of 25 g. of NH<sub>4</sub>VO<sub>3</sub>, 180 ml. of H<sub>2</sub>O and a g. of conc.  $H_{2}SO_{4}$  (see below) is prepared with stirring. The hot mixture is treated with SO<sub>2</sub> until a clear, dark-blue VOSO<sub>4</sub> solution is formed. Excess SO<sub>2</sub> is boiled off; the mixture is evaporated to 120 ml, and filtered. A porous clay cylinder 5 cm. in diameter and 10 cm. high, serving as diaphragm, is placed in a Pt cup (12 cm. in diameter and 6 cm. high). The vanadium salt solution is poured into the annular space; then, 25 ml. of 10% H<sub>2</sub>SO<sub>4</sub> and a Pt coil serving as anode are placed in the inner space, and the mixture is subjected to electrolysis for 45-50 min, at 3-4 v. and 6-7 amp. The electrolysis is continued until a pure green solution is obtained. The end-point of the reduction can be determined accurately by comparison with the color of a known vanadium (III) solution or by a control titration with KMnO<sub>4</sub>. The reduced solution is allowed to stand in a closed vessel. Crystallization of the alum is complete within 2-3 days the yield is 30-50%.

The  $H_2SO_4$  quantity *a* used initially determines whether the red or the blue alum form will be obtained. When a = 20 g., pure red crystals result, whereas when a = 40 g., the crystals are pure blue.

PROPERTIES:

Red or blue crystals; effloresce slowly in air with loss of water and oxidation. At 40-50°C, the alum melts in its water of crystallization, affording a green mass. Solubility (20°C): 40 g./100 g.  $H_2O.$  d 1.687.

### K, Rb AND Cs VANADIUM (III) ALUMS

The preparation of these compounds is similar to that of the NH<sub>4</sub> alum. The starting material is either  $V_2O_5$ , which is converted to a VOSO<sub>4</sub> solution by treatment with H<sub>2</sub>SO<sub>4</sub> and SO<sub>2</sub>, or a VOSO<sub>4</sub> compound. The stoichiometric quantity of K<sub>2</sub>SO<sub>4</sub>, Rb<sub>2</sub>SO<sub>4</sub> or Cs<sub>2</sub>SO<sub>4</sub> is added and the mixture is then electrolytically reduced. The ease and completeness of precipitation of these alums from their green solutions increase (and their solubility decreases) in the order  $K \rightarrow Rb \rightarrow Cs$ .

**REFERENCES:** 

A. Piccini. Z. anorg. Chem. <u>11</u>, 106(1896); <u>13</u>, 441 (1897); A. Bültemann. Z. Elektrochem. <u>10</u>, 141 (1904); J. Meyer and E. Markowicz. Z. anorg. allg. Chem. <u>157</u>, 211 (1926); H. Hartmann and H. L. Schäfer. Z. Naturforsch. <u>6a</u>, 754 (1951).

# Vanadium (IV) Oxysulfate

# (Vanadyl Sulfate)

### VOSO4

 $VOSO_4 \cdot 3 H_2O$ 

A solution of  $V_2O_5$  in pure sulfuric acid is reduced, preferably with  $SO_2$  (which is easier to work with than oxalic acid and ethanol because its excess may be readily removed).

$$V_2O_5 + H_2SO_4 + SO_2 + 5H_2O = 2 VOSO_4 \cdot 3H_2O$$
  
181.9 98.1 21.9 l. 90.1 434.1

Thus a stiff paste is prepared by stirring 190 g. of mildly calcined  $V_2O_5$  with 110 ml. of conc.  $H_2SO_4$  and 50 ml. of  $H_2O_5$ ; considerable heat is evolved during this operation. On the next day, 100 ml. of H<sub>2</sub>O is added and SO<sub>2</sub> introduced while heating the mixture on a water bath, until nearly all of the V<sub>2</sub>O<sub>5</sub> is dissolved. The dark-blue filtered solution is concentrated on a slowly boiling water (or steam) bath until a thick crystal mass is formed; the crystals are then suction-filtered and washed acid-free with 96% ethanol. The undesirable formation of a thick, blue sirup or a hard crystal cake, mentioned in the literature, seems to be due to impurities or to reaction conditions which differ from those given here; no such inconveniences are encountered when V<sub>2</sub>O<sub>5</sub>, prepared from thrice-recrystallized NH<sub>4</sub>VO<sub>3</sub>, and SO<sub>2</sub> as the reducing agent are used according to the above procedure. The bright, light-blue crystalline powder is dried over P<sub>2</sub>O<sub>5</sub> in a vacuum desiccator. Yield: 235 g. of VOSO<sub>4</sub> · 3 H<sub>2</sub>O. Evaporation of the mother liquor affords another crop of less pure vanadyl sulfate.

#### PROPERTIES:

Sky-blue crystalline powder. Readily soluble in  $H_2O$ , sparingly in ethanol. Hygroscopic; indefinitely stable in a closed bottle provided oxygen is absent.
REFERENCES

L. Vanino. Handb. d. präp. Chemie, Anorg. Teil [Handbook of Preparative Chemistry, Inorg. Part], 3rd ed., Stuttgart, 1925, p. 677; J. Dehnert. Thesis, Univ. of Jena, 1952, and a personal communication.

VOSO<sub>4</sub>, anhydrous

$$V_2O_5 + 2 H_2SO_4 = 2 VOSO_4 + 2 H_2O + 1/2 O_2$$
  
181.9 196.2 326.0 36.0

Analytical grade conc.  $H_2SO_4$  (100 ml.) and 3 g. of  $V_2O_5$  are boiled for several hours in a long-neck, round-bottom flask. The product is cooled (first in air, then in ice), and poured into 500-700 ml. of  $H_2O$ . The solid is suction-filtered until dry and washed with a large quantity of water. Since the product is still heterogeneous (yellow-brown particles, in addition to green ones), it is again subjected to the same treatment and, after suctionfiltration, dried with ethanol and ether, or over  $H_2SO_4$  in a desiccator.

PROPERTIES:

Green, loose, granular to finely crystalline powder; virtually insoluble in  $H_2O$ .

**REFERENCES**:

A. Sieverts and E. L. Müller. Z. anorg. allg. Chem. <u>173</u>, 313 (1928); V. Auger. Comptes Rendus Hebd. Séances Acad. Sci. <u>173</u>, 306 (1921).

## Vanadium Nitrides

VN

I. If pure V metal is available, it is best to proceed via the syntheses:

V +  $\frac{1}{2}$ N<sub>2</sub> = VN; V + NH<sub>3</sub> = VN +  $\frac{3}{2}$ H<sub>2</sub> 51.0 11.2 l. 65.0 51.0 22.4 l. 65.0

which give the purest products. Depending on the metal particle size, a reaction temperature between 900 and 1300°C is required. The starting material is placed in a boat (or crucible) made of  $Al_2O_3$  or Mo metal (see also the preparation of TiN, p. 1233, and of NbN, p. 1328). When other preparative methods are used, particularly those employing oxygenated starting materials (as in the

method described below), the nitride product inevitably contains some oxygen.

Alternate methods:

II. Very pure  $NH_4VO_3$  is heated for several hours at 900-1000°C in a very dry  $NH_3$  stream.

III.  $VOCl_3$  or  $V_2O_3$  is heated in an NH<sub>3</sub> stream.

IV.  $V_2O_3 + 3C + N_2 = 2VN + 3CO_1$ 

V. Deposition from gas phase on an incandescent wire; the gas contains  $VCl_4$ ,  $H_2$  and  $N_2$ . (see also TiN, p. 1233).

# V<sub>2</sub>N

I. Vanadium nitride, VN, is intimately mixed with the stoichiometric quantity of V metal powder; either the loose or the compacted mixture is then heated at 1100-1400 °C in an Al<sub>2</sub>O<sub>3</sub> or Mo crucible under Ar.

PROPERTIES:

Dark, submetallic materials.

VN: M.p. 2050°C; d 6.04. Homogeneity region:  $VN_{1.00}$ -  $VN_{0.71}$ . Crystal structure: B1 (NaCl) type.

 $V_2N$ : Homogeneity region:  $VN_{0.50}$ - $VN_{0.37}$ . Crystal structure: L 3 type.

REFERENCES:

- I, II. H. Hahn. Z. anorg. Chem. <u>258</u>, 58 (1949); V. E. Epelbaum and A. Brager. Acta Physicochim. URSS <u>13</u>, 595 (1940); W. D. Schnell. Thesis, Univ. of Freiburg i. Br., 1960.
- III. H. W. Roscoe. Ann. Pharm. Suppl. <u>6</u>, 114(1868); <u>7</u>, 191 (1870);
   N. W. Whitehouse. J. Soc. Chem. Ind. 27, 738 (1907).
- IV. E. Friederich and L. Sittig. Z. anorg. allg. Chem. <u>143</u>, 293 (1925).
- V. A. E. van Arkel and J. H. de Boer. Z. anorg. allg. Chem. <u>148</u>, 345 (1925); K. Moers. Z. anorg. allg. Chem. <u>198</u>, 243 (1931).

# Vanadium Phosphides

# VP<sub>2</sub>, VP, VP<sub><1</sub>

Vanadium phosphides are synthesized from the purest V metal available and P.

 $V + 2P = VP_{2i}$  V + P = VP51.0 62.0 113.0 51.0 31.0 82.0

Phosphorus and vanadium, the latter contained in an  $Al_2O_3$  crucible, are placed in a quartz tube of the type used for the "Faraday synthesis" (see p. 76 f.). The tube is thoroughly evacuated, melt-sealed and then heated for 24-48 hours in such a

way that the average temperature of the metal is 700-1000°C and that of the P is 480-550°C. To achieve homogeneous products and high P contents, the reaction must be carried out in stages, with intermediate grinding of the materials. Atmospheric oxygen must be carefully excluded during the grinding, to avoid appreciable absorption by the products.

The V-P system contains the phases  $VP_2$  and VP, as well as several phases in which the P content is low.

In the preparation of  $VP_2$ , an excess of P must be used from the very start because VP, once formed, reacts extremely slowly with additional phosphorus.

The vanadium phosphide VP can be obtained not only via the above synthesis, but also by thermal degradation of  $VP_2$  at 700-900°C (vacuum).

Lower phosphides (including  $V_3P$ ) are obtained by synthesis from the elements or from VP and V.

Alternate method: Electrolysis of  $V_2O_5$ -containing phosphate melts, with cathodic reduction to vanadium phosphides [M. Chène, Comptes Rendus Hebd. Séances Acad. Sci. 208, 1144 (1939); Ann. chimie [11], 15, 272 (1941)].

PROPERTIES:

Dark-gray substances; the lower phosphides have a submetallic luster. Not attacked by dilute  $H_2SO_4$ . Attacked by conc.  $H_2SO_4$  the more readily, the lower the phosphorus content. Incompletely soluble in nitric acid and aqua regia. Can be analyzed after decomposition by fusion with sodium carbonate-sodium nitrate.

VP: d 4.7; VPo.35: d 5.4.

REFERENCE:

M. Zumbusch and W. Biltz. Z. anorg. allg. Chem. 249, 1 (1942).

# Vanadium Carbides

## VC, V<sub>2</sub>C

I. It is probable that pure products can be obtained only by synthesis from the elements:

> V + C = VC;  $2V + C = V_2C$ 51.0 12.0 63.0 101.9 12.0 113.9

The reactants, preferably in finely subdivided form, are intimately mixed and, if needed, also compressed into pellets. The reaction is then carried out in a high vacuum. At a temperature of 1300°C, approximately 24 hours, or at 2000°C about 15 minutes, are required for homogenization of the product. The reaction is best carried out in a graphite crucible. Alternate methods: a) The elements are combined by heating in a carbon arc [A. Morette and M. G. Urbain, Comptes Rendus Hebd. Séances Acad. Sci. 202, 572 (1936)].

b) Vanadium oxides are mixed with carbon and heated in a  $H_2$  stream or in high vacuum. Carbides of an increased degree of purity are obtained if the final temperatures are allowed to reach 1700-2100°C (see the corresponding preparation of TiC, p. 1245 ff.) [C. Agte and K. Moers, Z. anorg. allg. Chem. <u>198</u>, 233 (1931); E. Friederich and L. Sittig, Z. anorg. allg. Chem. <u>144</u>, 169 (1925); A. Morette, Bull. Soc. Chim. France [5], <u>5</u>, 1063 (1938); W. Dawihl and W. Rix, Z. anorg. Chem. <u>244</u>, 191 (1940)].

c) Vapor deposition method (see TiC, p. 1246). An H<sub>2</sub> stream containing VCl<sub>4</sub> and toluene vapors is passed over an incandescent W wire [K. Moers, Z. anorg. allg. Chem. <u>198</u>, 243 (1931)].

PROPERTIES:

Dark, very hard, chemically resistant submetallic substances. The V-C system has two phases:

VC: Homogeneity region  $VC_{0.92}$ - $VC_{0.74}$ ; m.p. 2800°C. Crystal structure: B 1 (NaCl) type.

 $V_2C$ : Homogeneity region  $VC_{0.4}$ - $VC_{0.5}$ . Crystal structure: L 3 type.

**REFERENCES**:

A. Osawa and M. Oya. Sci. Rep. Tohoku Imp. Univ. <u>19</u>, 95 (1930); Chem. Zentr. <u>1930</u>, II, 298; W. Rostoker and A. Yamamoto. Trans. Amer. Soc. Metals <u>46</u>, 1136 (1954); N. Schönberg. Acta Chem. Scand. <u>8</u>, 624 (1954); M. A. Gurevich and B. F. Ormont. Zh. Neorg. Khimii <u>2</u>, 1566 (1957); W. D. Schnell. Thesis, Univ. of Freiburg i. Br., 1960.

#### Dibenzenevanadium (0)

#### $V(C_6H_6)_2$

$$\begin{array}{r} \mathrm{VCl}_{4} \ + \ \mathrm{Al} \ + \ 2 \ \mathrm{C}_{6} \mathrm{H}_{6} \ = \ \mathrm{V}(\mathrm{C}_{6} \mathrm{H}_{6})_{2} \ \mathrm{AlCl}_{4} \\ \mathrm{192.8} \ 27.0 \ 156.2 \ 376.0 \\ 5 \ \mathrm{V}(\mathrm{C}_{6} \mathrm{H}_{6})_{2}^{+} \ + \ 8 \ \mathrm{OH}^{-} = \ 4 \ \mathrm{V}(\mathrm{C}_{6} \mathrm{H}_{6})_{2} \ + \ \mathrm{VO}_{4}^{3-} \ + \ 4 \ \mathrm{H}_{2} \mathrm{O} \ + \ 2 \ \mathrm{C}_{6} \mathrm{H}_{6} \end{array}$$

The reactor is a 250-ml., three-neck flask equipped with an agitator, a reflux condenser and a mercury safety valve. A mixture

of 10 g. (0.37 moles) of dry Al powder and 4 g. (0.03 moles) of finely subdivided AlCl<sub>3</sub> and 150 ml. of absolute benzene (an excess) is added. The system is purged by introducing N<sub>2</sub> via the reflux condenser. The flask is equipped with a pressure-equalizing dropping funnel containing a solution of 9 g. (5 ml., 0.047 moles) of freshly distilled VCl<sub>4</sub> in 50 ml. of benzene. The flask contents are heated to a boil while stirring under a blanket of N<sub>2</sub>. The VCl<sub>4</sub> solution is then added dropwise (slowly) over a period of one hour, and the mixture is agitated and boiled for an additional 20 hours. It assumes a golden yellow color. It is allowed to cool, and the dropping funnel, the agitator and the reflux condenser are replaced (under a N<sub>2</sub> stream) with stoppers and a vacuum adapter. The benzene is then removed in vacuum, with heating toward the end of the distillation. The dry residue is reduced to small pieces (in the same flask and under  $N_2$ ). While protecting it from air, a part of the residue is then transferred to a 500-ml. separatory funnel kept under N<sub>2</sub> (which is introduced through a side tube), and covered with 200 ml. of N<sub>2</sub>-saturated petroleum ether. This is followed by repeated additions, with vigorous shaking, of 100-ml. portions of N<sub>2</sub>-saturated 1 N NaOH. After complete hydrolysis, the mixture is allowed to stand, the aqueous layer is separated, and the brown-red petroleum ether solution is washed three times (in the absence of air) with 20-ml. portions of N<sub>2</sub>-saturated H<sub>2</sub>O. Hydrolysis of the remainder of the solid reaction product is carried out similarly, in 2-3 operations. The combined petroleum ether extracts are dried for 15 minutes over solid KOH and the solvent is evaporated in vacuum. The residue is sublimed at 120-150°C in high vacuum, placed in a V-shaped washing tube, washed three times with 5-10 ml, of air-free, absolute petroleum ether to remove organic impurities, and finally resublimed. Yield: 1.3-2.5 g. (13-25%, based on VCl<sub>4</sub>).

#### **PROPERTIES**:

Formula weight 207.18. Brown-red to black crystalline substance. M.p. (in N<sub>2</sub>) 277°C. Sublimes in high vacuum at 120-150°C, decomposes above 300°C. Instantly oxidized by air (decomposition). Soluble in benzene, ether, pyridine, petroleum ether and acetone; the solutions are brown-red and stable in the absence of air. Insoluble or only sparingly soluble in CCl<sub>4</sub> and methanol. Not dissolved or attacked by  $H_2O$  in the absence of  $O_2$ , but decomposed in the presence of air.

REFERENCE:

E. O. Fischer and H. P. Kögler. Chem. Ber. 90, 250 (1957).

## Potassium Hexathiocyanatovanadate (III)

#### K<sub>3</sub>V(SCN)<sub>6</sub>

A vanadium (III) solution is obtained from  $V_2O_5$  by reduction of the latter with  $SO_2$ , followed by electrolysis. This solution is then reacted with KSCN:

$$\begin{array}{rrrr} V_2O_5 &+& 5\ H_2SO_4 &+& 12\ KSCN &+& 4\ e^-\\ 181.9 && 490.4 && 1166.2\\ &=& 2\ K_3V(SCN)_6 &+& 3\ K_2SO_4 &+& 5\ H_2O &+& 2\ SO_4^{2^-}\\ 1033.5 && 522.8 && 90.1 \end{array}$$

Fine  $V_2O_5$  powder (91 g., 0.5 moles) is stirred with 250 ml. of 4 N H<sub>2</sub>SO<sub>4</sub>, the suspension heated, and SO<sub>2</sub> introduced until a clear, pure blue solution is obtained. The mixture is heated to a boil to remove the excess SO<sub>2</sub> and is then concentrated to 2/3 of its previous volume.

This solution is subjected to electrolytic reduction in a cell containing a clay cylinder diaphragm; the current is 2-3 amp. at 10 v. (the procedures are those described on pp. 1277 and 1284). The electrolysis is continued until the electrolyte at the cathode shows the pure green color of vanadium (III). The best electrodes are those made of platinum sheet.

The theoretical quantity of KSCN used depends on the volume of the cathode electrolyte and is calculated on the assumption that 6 moles (or 583 g. of KSCN) corresponds to 1 g.-atom of vanadium. This quantity of KSCN, in the form of a concentrated aqueous solution, is then added to the above electrolyte. The resulting red liquid is concentrated on a water bath; the residue is dissolved in the minimum amount of ethanol and treated with ether until  $K_2SO_4$ no longer precipitates. The  $K_2SO_4$  is filtered off, the filtrate is evaporated on a water bath, and the precipitation operation is repeated. The residue thus obtained (it is completely free of  $K_2SO_4$ ) is recrystallized from a small amount of  $H_2O$ . Wellformed crystals of the dihydrate,  $K_3V(SCN)_{\rm g} \cdot 2 H_2O$ , are obtained.

The anhydrous salt is obtained by drying the dihydrate over  $H_2SO_4$  in a vacuum desiccator, finely pulverizing it, and then completely dehydrating it under vacuum (drying pistol) at 95°C until constant weight is reached.

PROPERTIES:

 $K_3V(SCN)_6 \cdot 2 H_2O$ : Formula weight 552.79. Brown-red, leaf-like crystals.

K<sub>3</sub>V(SCN)<sub>6</sub>: Formula weight 516.76. Very hygroscopic.

REFERENCE:

O. Schmitz-Dumont and G. Broja. Z. anorg. Chem. 255, 299 (1948).

# Niobium Metal, Tantalum Metal

Because of the tendency of Nb and Ta metal to form very stable oxides, nitrides and carbides, the difficulties in the preparation of these metals are similar to those encountered in the preparation of Ti and Zr (see pp. 1161 and 1172).

Three methods are available for industrial preparation of the pure metals. The first involves electrolysis of fluoride melts containing  $K_2NbOF_5$  (or  $K_2TaF_7$ ), as well as a certain amount of the corresponding oxides,  $Nb_2O_5$  or  $Ta_2O_5$ . An iron fusion pot serves as the cathode and graphite rods as the anodes. The resulting metal is a fine powder which may be separated from the admixed salt melt by a variety of processes. In the second method an oxide and a carbide, for instance  $Nb_2O_5 + 5$  NbC, are mixed, compressed into pellets and heated in high vacuum to temperatures exceeding 1600°C. In the third method, a double fluoride is reduced either with liquid sodium or sodium vapor. In each case, the material is processed further via powder-metallurgical methods, by subjecting it to repeated and alternating procedures which increase density and hot degassing treatments.

## I. REDUCTION WITH SODIUM OR CALCIUM

The laboratory preparation of the metal powder proceeds via reduction of the halides with sodium, calcium or  $CaH_2$ .

Thus, for example, 50 g. of high-purity, dry  $K_2TaF_7$  and 18 g. of Na (precut into small pieces under benzene) are placed in a heavy-wall steel vessel, tightly closed off with a well-fitting conical lid, which is fastened on with screws. The system is heated for one hour at red heat, allowed to cool completely, and reopened; the reaction mixture, which still contains some free Na, is carefully introduced, in small portions and with agitation, into 500 ml. of H<sub>2</sub>O. The lumps disintegrate, and the resulting metal powder is treated several times with H<sub>2</sub>O, then hot nitric acid (d 1.2), strong hydrochloric acid (1:1) and, finally, again and thoroughly with water. It is then dried.

According to Kroll, the oxides can be reduced with calcium metal in the presence of  $CaCl_2$  as the fluxing agent

$Nb_2O_5$	+	5 Ca	=	$2 \mathrm{Nb}$	+	5 CaO
265.8		200.4		185.8		280.4
$Ta_2O_5$	+	5 Ca	=	2 Ta	+	5 CaO
441.9		200.4		361.9		280.4

Redistilled Ca turnings of the best grade are used in approximately 30% excess. The presence of  $CaCl_2$  is essential. For example, in a preparation of a small amount of metal, 8 g. of Nb<sub>2</sub>O<sub>5</sub> (or 13 g. of Ta<sub>2</sub>O<sub>5</sub>), 8 g. of Ca and 15 g. of CaCl<sub>2</sub> are placed in a heavy-wall tubular iron crucible (25 mm. in diameter, 80 mm. long) which is filled with Ar and made gas-tight by closing it off with a welded-on (oxygen-acetylene flame) iron plug. This crucible is then heated for half and hour at about 1000-1100°C, allowed to cool completely, and sawed open. The contents are treated with water and acids, as described above. The metal powder contains some very fine particles which are best separated by decantation and discarded.

The above two methods are equally applicable to Nb and to Ta. Various modifications of these methods are possible; in particular, the reduction with Ca may be replaced by one with CaH<sub>2</sub> [see the corresponding procedures for Ti, methods I and II, pp. 1161-1165, as well as G. Tourné, Ann. Chim. [13],  $\underline{4}$ , 949 (1959)].

The metal powders thus obtained are not particularly pure and often contain not more than 97% of the metal, which is accompanied by hydrogen, some oxygen, and sometimes also small amounts of nitrogen, carbon and iron. The purity can be increased by repeating the treatment with the reducing metal or with  $CaH_2$  and, also, by increasing the batch size. At any rate, it is of advantage to purify these powders further by heating them at a high temperature under vacuum, for example, via procedure II. However, the further conversion of Nb or Ta powder to the corresponding halides is not affected by the impurities, provided they are not metallic.

## **II.** PURIFICATION BY CALCINATION

Low-purity Nb or Ta can be freed of most of its contaminants by heating to red heat in a vacuum. Both Nb and Ta have very high melting points, and thus the contaminants can simply be evaporated. In this procedure, the metal powder, pressed into oblong rods and clamped between water-cooled molybdenum jaws, is resistanceheated with a high current, or the loose or compressed metal powder is heated to red heat on a support of ThO<sub>2</sub>, W or Ta sheet, placed in a tungsten electrical heating element. The best type of heater is the furnance shown in Part I, p. 40 f., wherein a tungsten tube or a tungsten trough is used as the heat conductor. Highfrequency induction heating can also be used. In each case, a very high vacuum of at least  $10^{-5}$  mm, is of controlling importance, if the purification is to be efficient. The material is first degassed by preheating it for about 1 hour at 1200°C. The temperature is then slowly increased and then maintained at 2000-2200°C for some time (one to several hours). If this temperature is reached too quickly or if the vacuum is not good, no purification will be achieved; in addition, in the case of Nb, there will also be undesirable melting, among other things (formation of a eutectic between metal and impurities).

## III. CRYSTAL-GROWING (OR VAPOR-DEPOSITION) PROCESS

High-purity Nb or Ta can be obtained by deposition on an incandescent wire from gaseous NbCl<sub>5</sub> and TaCl<sub>5</sub>, either in the presence or the absence of hydrogen. This crystal-growing process corresponds closely to that described for titanium on p. 1168 ff., particularly as far as the apparatus is concerned. Since tungsten and Nb or Ta readily form brittle alloys, a tungsten nucleating (substratum) wire cannot be used in this case; instead, one uses an approximately 0.1-mm.  $\phi$  wire of the metal to be deposited. In addition, Ni (and not W) terminals are used, and the system is degassed by heating to red heat in vacuum before the start of the run. The chloride (NbCl<sub>5</sub> or TaCl<sub>5</sub>) is introduced into the side tube and vacuum-sublimed in situ; the entire reactor system is heated and thoroughly degassed prior to sublimation, because the absence of gas is essential to the quality of the deposited metals. The reactor remains connected to the vacuum pump throughout the entire process. Vapor deposition takes place by heating the chloride and the entire vessel to about 100°C; the substratum wire is heated to 1800°C in the case of Nb, and to 2000°C in the case of Ta. The deposited metals are of very high purity.

The thickness of the incandescent wire changes continuously during the reaction, so that careful supervision of the process and good electrical control are imperative. This disadvantage is circumvented in Rolsten's modification of the process, whereby the volatile iodide of the metal is decomposed at 750-1100°C in an indirectly heated fused quartz (or Vycor) tube.

Alternate methods: a) Reaction of the chlorides with Mg [J. Prieto, A. J. Shaler and J. Wulff, Metals Technol. <u>14</u>, No. 6 (1947)].

b) Reduction of the oxides with Si while volatilizing the nascent SiO [E. Zintl et al., Z. anorg. allg. Chem. <u>245</u>, 1 (1940)].

## IV. COMMINUTION OF THE SOLID METAL

Commercially available solid Nb and Ta (sheet, wire, etc.) are usually far purer than the powdered material. When metal powder of very highest purity is required for the preparation of a Nb or Ta compound, solid waste pieces may be used to advantage. They are pulverized by hydrogenation at 500-600°C (see hydrides) and cooling under  $H_2$ . The resulting hydrides are very brittle and are readily pulverized to the desired size. The powder thus obtained is then dehydrogenated at 1000°C in an extremely high vacuum. The decrease in purity occurring during these operations is negligibly small provided very pure  $H_2$  is used, the pulverization of the hydrides is carried out in an inert gas atmosphere, and the heating and degassing of the material is always carried out so slowly that no appreciably loss of vacuum occurs in the system (which is permanently connected to a vacuum pump).

PROPERTIES:

Nb: Atomic weight 92.91. M.p. 2468°C; d 8.58.

Te: Atomic weight 180.95. M.p. 3030°C; d 16.6.

These two metals are not attacked by mineral acids (with the exception of hydrofluoric); they are readily soluble in a mixture of concentrated hydrofluoric and nitric acids. Crystal structure: A 2 (W) type.

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  - I. K. R. Krishnaswami. J. Chem. Soc. (London) <u>1930</u>, 1277; W. Kroll. Z. anorg. allg. Chem. <u>234</u>, 42 (1937); J. W. Marden and M. N. Rich. U. S. Patent 1,728,941 (1927/29); G. Brauer. Unpublished experiments, Darmstadt, 1942; W. E. Dennis and A. F. Adamson. U.K.A.E.A. Techn. Note No. 92 (1954); E. F. Block. U.S. Bureau of Mines, paper given at the Achema meeting, Frankfurt, 1958.
- II. H. Bückle. Z. Metallkunde <u>37</u> (Metallforschg. 1), 53 (1946);
  R. H. Myers. Metallurgia (Manchester) <u>38</u>, 307 (1948); Symposium on the Metallurgy of Niobium, J. Inst. Metals <u>85</u> (1956-57);
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- III. W. G. Burgers and J. C. M. Basart. Z. anorg. allg. Chem. <u>216</u>, 223 (1934); see also the literature cited in section on Ti, method V; R. F. Rolsten. Trans. A.I.M.E. <u>215</u>, 472 (1959); J. Electrochem. Soc. <u>106</u>, 975 (1959); Z. anorg. allg. Chem. <u>305</u>, 25 (1960).

# Vanadium, Niobium and Tantalum Hydrides

After thorough degassing at red heat in vacuum, the metals are heated in an atmosphere of extremely pure  $H_2$ . The rate of hydrogen absorption depends strongly on the particle size of the metals, the metal purity and the pretreatment method. Above 400°C, even solid Nb and Ta (sheet, wire, etc.) react fairly rapidly. The rate of the reaction increases with the metal purity and is particularly high if the metal history includes a previous hydrogenation and dehydrogenation. In that case, hydrogen is sometimes absorbed even at room temperature. Both the NbH and the TaH systems may exist in two stable phases; the transitions from one to another, however, are not clearly reflected in the isotherms and isobars of the H<sub>2</sub> equilibrium pressure. Hence, depending on the temperature and equilibrium H<sub>2</sub> pressure, the hydrogenated material may contain hydrogen in all ratios up to the limiting composition, which is approximately NbH<sub>0.93</sub> and TaH<sub>0.8</sub> (corresponding to 112 ml. of H<sub>2</sub>/g. of Nb and 56 ml. of H<sub>2</sub>/g. of Ta. One gram of V absorbs a maximum of 205 ml. of H<sub>2</sub>, corresponding to the limiting formula VH<sub>0.94</sub>.

The deuterides of Nb and Ta are analogous to the hydrides prepared under the same conditions.

PROPERTIES:

Lustrous metallic or metallic gray appearance, much like that of the free metals. The lower hydrides (up to approximately  $MH_{0.1}$ ) are quite hard but are still ductile, increasing in brittleness with increasing H content and becoming extremely brittle at high hydrogen ratios. The hydrogen can be removed in a high vacuum at temperatures exceeding 400°C, and rapidly between 800 and 1000°C.

REFERENCES:

General: D. P. Smith. Hydrogen in Metals, Chicago, 1948.

- V-H: L. Kirschfeld and A. Sieverts. Z. Elektrochem. <u>36</u>, 123 (1930); H. Huber, L. Kirschfeld and A. Sieverts. Ber. dtsch. chem. Ges. <u>59</u>, 2891 (1926); M. J. Trzeciak, D. F. Dilthey and M. W. Mallett. Batelle Mem. Inst. Rep. 1112 (1956).
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- Ta-H: A. Sieverts and H. Brüning. Z. phys. Chem. (A) <u>174</u>, 365 (1935); A. Sieverts and E. Bergner. Ber. dtsch. chem. Ges. <u>44</u>, 2394 (1911).

# Niobium (II) Chloride

NbCl<sub>2</sub>

 $\begin{array}{rrrr} Nb &+& 2 \ NbCl_3 &=& 3 \ NbCl_2 \\ 92.9 & & 398.6 & & 491.5 \end{array}$ 

Stoichiometric quantities of Nb metal powder and NbCl<sub>3</sub> are weighed under anhydrous conditions, triturated and placed in a quartz tube which is closed at one end. Constrictions b and d(Fig. 305) are then made in the end of the tube and it is connected to a high-vacuum source. (It should be remembered in calculating the amount of NbCl<sub>3</sub> that this compound exhibits a considerable phase width and can, therefore, be of varying composition. The use of NbCl<sub>2.67</sub> is particularly convenient.)

To degas the contents, the reactor is heated for 12 hours in high vacuum at 200°C and is then melt-sealed at constriction d. The reaction is completed by heating the entire tube at 800°C for two to three days. The tube is then chilled in water and the fairly volatile by-products (NbCl<sub>5</sub>, NbOCl<sub>3</sub>) are distilled forward into tube section c by establishing a 200/20°C temperature gradient. The NbCl<sub>2</sub> remains in a.



Fig. 305. Preparation of niobium (II) chloride. The quartz reactor is 8 mm. I.D. Length: a = 50 mm., c = 20 mm.

PROPERTIES:

Formula weight 163.82. Black-brown crystals. Stable in air, insoluble in  $H_2O$  and organic solvents. When heated in an evacuated tube, the NbCl<sub>2</sub> decomposes at a temperature gradient of 800/550 °C via the equilibrium reaction:  $4 \text{ NbCl}_2 = \text{Nb} + 3 \text{ NbCl}_{2.67}$ ; at a temperature gradient greater than 800/20 °C, the reaction is:  $2 \text{ NbCl}_2 = \text{Nb} + \text{NbCl}_4$ . Heating in air produces NbOCl<sub>3</sub> and Nb<sub>2</sub>O<sub>5</sub>.

**REFERENCES:** 

H. Schäfer and K. D. Dohmann. Z. anorg. allg. Chem. <u>300</u>, 1 (1959);
H. Schäfer and F. Kahlenberg. Z. anorg. allg. Chem. <u>305</u>, 291 (1960).

#### Niobium (III) Chloride

NbCl<sub>3</sub>

I.

High purity, oxygen-free hydrogen is passed through a vessel containing NbCl<sub>5</sub> heated to 150-190°C. The gas stream then passes through a Pyrex or Vycor tube heated to 400-530°C and solid precipitate of green-black NbCl<sub>3</sub> deposits on the walls. Complete conversion of NbCl<sub>5</sub> to NbCl<sub>3</sub> may be achieved, provided the gas rate is low. The lower partial pressures of NbCl<sub>5</sub> (at a saturation temperature of 150°C) and the higher temperature of the reaction zone (that is, 530°C) lead to a limiting composition which is low in chlorine (NbCl<sub>2,67</sub>).

#### G. BRAUER

Brubaker and Young prepared NbCl<sub>5</sub> from Nb metal and Cl<sub>2</sub> in the apparatus shown in Fig. 310; they then allowed it to sublime in an H<sub>2</sub> stream through constriction c into the right-hand tube, which was heated to 500°C and equipped with a cold finger. The NbCl<sub>3</sub> separated both as a dark crust on the tube wall and as a coneshaped deposit on the cold finger. The product may be pyrophoric when prepared by this procedure; it should, therefore, be handled only under a protective N<sub>2</sub> blanket.

II.

$$3 \text{ NbCl}_{5} + 2 \text{ Nb} = 5 \text{ NbCl}_{3}$$
  
 $810.6 185.8 996.4$ 

High-surface Nb metal (e.g., foil) and a slight excess of NbCl<sub>5</sub> are placed in an evacuated reactor tube which is then sealed. (The NbCl<sub>5</sub> can be prepared in the reactor itself before introducing the Nb metal; this can be accomplished by reacting weighed amounts of Nb  $_{2}O_{5}$  and CCl<sub>4</sub>—see preparation of NbCl<sub>5</sub>, method III.) The sealed horizontal reactor is heated for three days in a temperature gradient such that the end of the tube containing Nb is at 390°C, while the remainder of the tube is at 355°C. The contents are thus converted to NbCl<sub>3</sub>, which, as a result of the reversible equilibrium  $NbCl_3$  (solid) +  $NbCl_5$  (gas) =  $2NbCl_4$  (gas), is transported into the 355 °C zone where it deposits as crystals. It can then be resublimed by reversing the temperature gradient. At the end of the procedure, only the part of the tube containing the NbCl<sub>3</sub> is heated for a few minutes to 390°C while keeping the other end at 20°C, thus driving the NbCl<sub>5</sub> to the cold end. The tube is allowed to cool and is then opened under anhydrous conditions.

The partial pressure of NbCl<sub>5</sub> and, hence, the composition of the NbCl<sub>3</sub> phase can be varied in this synthesis by varying the ratio of the NbCl<sub>5</sub> to the volume of the sealed tube (saturation pressure of NbCl<sub>5</sub> at  $355^{\circ}$ C is 8 atm.).

III.  $3 \text{ NbCl}_{5} + 2 \text{ Al} = 3 \text{ NbCl}_{3} + 2 \text{ AlCl}_{3}$  $810.6 \quad 53.9 \quad 597.8 \quad 266.7$ 

Sublimed NbCl<sub>5</sub> is heated with a less than stoichiometric quantity of Al powder in an evacuated, sealed tubular reactor. For example, 1.2 g. of NbCl<sub>5</sub> and 0.08 g. of Al may be used. The entire length of the tube is heated to 275 °C for about 40 hours. It is then placed in a temperature gradient, with the main section encased in an aluminum block at 300 °C and the protruding end at room temperature. The partially formed NbCl<sub>4</sub> decomposes into NbCl<sub>3</sub> and NbCl<sub>5</sub>, the excess NbCl<sub>5</sub> and AlCl<sub>3</sub> sublime into the tube end, while the green-black NbCl<sub>3</sub> remains in the main (lower) section of the tube. It is recovered by opening the tube; no special precautions against air are needed. The product of this process usually contains a small amount of Al<sub>2</sub>O<sub>3</sub> (about 0.7%) which is introduced with the Al powder.

1298

Alternate methods: a) Reduction of NbCl<sub>5</sub> with activated (excitation)  $H_2$  at 200°C [V. Gutmann and H. Tannenberger, Monatsh. Chem. 87, 769 (1956)].

b) Preparation from Nb metal in an HCl stream at  $300^{\circ}$ C. A mixture of NbCl<sub>3</sub> and NbCl<sub>5</sub> is obtained [V. Y. Spitsyn and N. A. Preobrazhenskiy, Zh. Obshch. Khimii <u>10</u>, 785 (1940); C. H. Brubaker and R. C. Young (1951)].

PROPERTIES:

Green-black; crystallizes in crusts, rods, or plates. Under sufficiently high NbCl<sub>5</sub> pressure and in the absence of air, NbCl<sub>3</sub> is stable at 800°C. It disproportionates to NbCl<sub>5</sub> and Nb in a temperature gradient. Only slightly air-sensitive at room temperature. Insoluble in  $H_2O$ , dilute acids and dilute alkali. Attacked by oxidizing agents at varying rates depending on the concentration and the temperature. Insoluble in organic solvents, even in ethanol.

Exhibits a rather wide homogeneity region (between NbCl<sub>3.13</sub> and NbCl<sub>2.67</sub>). d 3.75.

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## Niobium (IV) Chloride

## NbCl<sub>4</sub>

Prepared by reduction of NbCl<sub>5</sub>.

I.

 $\begin{array}{rrrrr} 2\,\mathrm{NbCl}_5 &+ \mathrm{Fe} &= 2\,\mathrm{NbCl}_4 &+ \mathrm{FeCl}_2 \\ 540.4 & 55.9 & 469.5 & 126.8 \end{array}$ 

The reaction is carried out in a sealed tube divided into two sections by a constriction, as shown in Fig. 306. The NbCl<sub>5</sub> (3 g.) is introduced into the closed tube end a by connecting the tube via its still open end b to the apparatus used for isolation of NbCl<sub>5</sub> (Fig. 309). Pure iron (0.24 g., e.g., Armco iron turnings or reduced iron) is placed in section b, the tube is drawn out to a

point, high vacuum is applied, and the tube is sealed at the point. The horizontal sealed tube is encased in two closely spaced aluminum blocks which are electrically heated to two different temperatures. The section containing NbCl<sub>5</sub> is heated to 195°C, and that with Fe to 400°C. The reaction time is at least 40 hours. The gaseous NbCl<sub>5</sub> is reduced and NbCl<sub>4</sub> separates as well-formed crystals in a transition region between the two temperature zones. The FeCl<sub>2</sub> (which at 400°C is still not very volatile) and unreacted NbCl<sub>5</sub> are found in the other sections of the tube.



Fig. 306. Preparation of niobium (IV) chloride.

To prevent scattering of the reaction products by a rapid influx of gas (dry air,  $CO_2$  or  $N_2$ ) while opening the evacuated tube, the tip of the tube should be scratched, placed in a slightly larger vacuum hose, and broken off under vacuum. The tube may then be gradually filled with gas through the vacuum hose. To isolate the NbCl<sub>4</sub>, the tube is then broken at an appropriate spot.

II. The NbCl<sub>5</sub> can also be reduced with Nb metal.

$$\begin{array}{rrrr} 4 \, NbCl_{5} \, + \, Nb \, = \, 5 \, NbCl_{4} \\ 1080.8 & 92.9 & 1173.7 \end{array}$$

Thus,  $NbCl_5$  and an excess of Nb metal are sealed into a tube described in method I and heated in the same temperature gradient; the reaction is complete in about 16 hours.

Alternate methods: a) Reduction of  $NbCl_5$  with Al metal; requires a subsequent distillation of excess  $NbCl_5$  and  $AlCl_3$ .

b) Reduction of  $NbCl_5$  with  $H_2$  at 2 atm. (generated when the tube is filled at STP, sealed and then heated). The reaction does not go to completion and  $NbCl_5$  and  $NbCl_4$  must be separated by sublimation.

c) Reaction of NbCl<sub>5</sub> and NbCl<sub>3</sub>; as in most methods, an excess of NbCl<sub>5</sub> is usually required to depress the decomposition of NbCl<sub>4</sub>.

PROPERTIES:

Brown-black crystal needles; pure brown in transmitted light. Sublimable at about 275 °C, provided decomposition into NbCl<sub>5</sub> and

NbCl<sub>3</sub> is prevented by a sufficiently high NbCl<sub>5</sub> pressure. Decomposes on exposure to air and moisture (color change first to black, then to white). Dissolves in a small amount of  $H_2O$  and in dilute hydrochloric acid, giving a dark-blue solution.

REFERENCES:

H. Schäfer, C. Göser and L. Bayer. Z. anorg. allg. Chem. <u>265</u>, 258 (1951); data from the Chemical Laboratory of the University, Freiburg i. Br., 1952; H. Schäfer, L. Bayer and H. Lehmann. Z. anorg. allg. Chem. <u>268</u>, 268 (1952).

# Tantalum (IV) Chloride TaCl<sub>4</sub>

I.

 $\begin{array}{rrrr} 4\,{\rm TaCl_5} &+ {\rm Ta} = 5\,{\rm TaCl_4} \\ 1433.0 & 181.0 & 1614.0 \end{array}$ 

A quartz reactor tube with a narrow hooked constriction, shown in Fig. 307 (I), is thoroughly degassed by heating in a high vacuum; then, 4 g. of Ta metal (preferably foil) and 10-15 g. of TaCl<sub>5</sub> are introduced into the tube on opposite sides of the constriction, and the tube is sealed under high vacuum. It is then heated in a slanted position in a temperature gradient so that the liquid TaCl<sub>5</sub> (in the higher end of the tube) is at 280°C and the Ta at 630°C. The nascent



Fig. 307. Preparation of tantalum (IV) chloride. b) aluminum foil; f) TaCl<sub>5</sub>.

 $TaCl_4$  deposits in the 280° zone (large crystals) but is separate from the TaCl<sub>5</sub>. A six-day runyields 8-10g.; there is also a residue of unreacted starting materials. Before opening the tube, the reactor is cooled, and the section containing TaCl<sub>4</sub> is reheated to 200°C to separate any admixed TaCl<sub>5</sub> by sublimation. The opening of the tube and handling of TaCl<sub>4</sub> should be carried out in the absence of moisture. G. BRAUER

II. 
$$3 \operatorname{TaCl}_5 + Al = 3 \operatorname{TaCl}_4 + AlCl_3 \\ 1074.7 \quad 27.0 \quad 968.3 \quad 133.4$$

The reaction is carried out in a sealed reactor tube (Fig. 307, II) in a high vacuum. Aluminum foil (for example, 50 mg.) is introduced at b, while TaCl<sub>5</sub> (4-5 g.) in an ampoule is at f; both are introduced under anhydrous conditions. The TaCl<sub>5</sub> is made to sublime (in high vacuum) toward tube section d, and the tube is sealed off at constrictions a and e. The sealed tube is heated for 70 hours in a temperature gradient (see Fig. 306) such that b is at 400°C and the remainder of the tube at 200°C. The TaCl<sub>4</sub> deposits at c as large crystals. The tube is allowed to cool and only section c is reheated to 200°C to remove any TaCl<sub>5</sub> present in it.

III. 
$$\begin{array}{rcl} T_aCl_5 + H &= T_aCl_4 + HCl\\ 358.2 & 322.8 \end{array}$$

Whereas  $TaCl_5$  reacts with molecular  $H_2$  only at temperatures exceeding 500°C (to form Ta metal), the reduction of  $TaCl_4$  with  $H_2$  activated by a high-frequency electrical discharge can be carried out at 200°C. The apparatus is the same as that for the preparation of  $TaBr_4$  (Fig. 311) and the process is the same in all its details. A two-hour run completely reduces 1 g. of  $TaCl_5$ .

PROPERTIES:

Brown-black crystals. Moisture-sensitive; decomposes with oxidation on exposure to air. On heating in vacuum, disproportionates to  $TaCl_5$  and a lower chloride; on heating in air forms  $Ta_2O_5$ and volatile  $TaCl_5$ . Partly soluble in  $H_2O$  and dilute acids, yielding coffee-brown solutions; an insoluble dark material is also formed.

**REFERENCES:** 

- I. H. Schäfer and F. Kahlenberg. Z. anorg. allg. Chem. 305, 178 (1960).
- II. H. Schäfer and L. Grau. Z. anorg. allg. Chem. 275, 198 (1954).
- III. V. Gutmann and H. Tannenberger. Monatsh. Chem. <u>87</u>, 769 (1957).

# Niobium (V) and Tantalum (V) Chlorides

NbCl<sub>5</sub>, TaCl<sub>5</sub>

NbCl₅

I.

$$\begin{array}{rcl} {\rm Nb} &+& {}^{5/}_{2}\,{\rm Cl}_{2} &=& {\rm NbCl}_{5} \\ {\rm 92.9} && 56\ l. && 270.2 \end{array}$$

Niobium metal, either as a powder or as a solid, can be readily chlorinated in a  $Cl_2$  stream. The reaction is best carried out in a

tube similar to that shown in Fig. 312 (preparation of NbBr<sub>5</sub>); however, the saturation tube is replaced by a T connector through which the sealed tube can be evacuated or dry  $N_2$  or  $Cl_2$  introduced. The end arrangement of the apparatus varies depending on the expected amount of NbCl<sub>5</sub>. Air must be carefully displaced by evacuation or purging with  $N_2$ . The reaction with  $Cl_2$  starts at 125-240°C; at 240°C, it takes only a few fours regardless of the Nb particle size. The absorption of  $Cl_2$  is usually quite rapid. The NbCl<sub>5</sub> product is taken out under anhydrous conditions and resublimed in an appropriate manner (the apparatus of Fig. 308 can be used).

II.

$$Nb_2O_5 + 5 SOCl_2 = 2 NbCl_5 + 5 SO_2$$
  
265.8 594.9 540.4 320.3

A common, carefully dried bomb tube is charged with 2.7 g. of Nb<sub>2</sub>O<sub>5</sub> and 10 ml. of SOCl<sub>2</sub>. Care should be taken in the preparation of Nb<sub>2</sub>O<sub>5</sub> (from precipitated hydrated oxide) not to exceed 400°C, since excessively calcined oxide is inactive and reacts incompletely.



Fig. 308. Resublimation of niobium and tantalum pentahalides under anhydrous conditions. Thus, if the oxide is excessively calcined, it is fused with KHSO<sub>4</sub>, the melt hydrolyzed, the hydrated oxide precipitated with ammonia and then dried for a long time at about 400°C. Before use, the SOCl<sub>2</sub> is purified by first refluxing it for 4 hours in the presence of S and then fractionally distilling it in a column [D. L. Cottlet, J. Amer. Chem. Soc. <u>68</u>, 1380 (1946)].

The filled and sealed tube is heated for 3 hours at 200°C. On slow cooling, NbCl<sub>5</sub> crystallizes in needles. The tube is cooled to --10°C, opened, and the SO<sub>2</sub> discharged by heating to room temperature; the excess SOCl<sub>2</sub> is removed by further slight heating in vacuum. To achieve this, as well as for the further handling of NbCl<sub>5</sub>,

the apparatus shown in Fig. 309 is attached to the open bomb. Because of the high sensitivity to moisture exhibited by NbCl<sub>5</sub>, it is absolutely necessary to equip the apparatus with devices which permit handling of the product in such a way that even traces of moisture will be excluded. The NbCl<sub>5</sub> remaining at *a* is first moved to *b* by subliming it under vacuum; it is then transferred to *g* (under nitrogen) for further handling. This is achieved by removing the ground cap *c*, stretching a thin perforated rubber cap over the tube and introducing through this cap a small spatula with a long handle. Tube g of Fig. 309 is used for storing the product; it is closed at f with a ground glass stopper and permits partial removal of the chloride. Protection from moisture is provided by N<sub>2</sub>, which is introduced through h. Naturally, other types of containers can be used instead of g, for example, a simple ampoule which is sealed off.



Fig. 309. Purification by sublimation and filling of a vessel with niobium (V) chloride. a, b) bomb tube; c, d) openings for introducing spatulas and long-handle hooks; g) storage vessel.

III. The oxide can also be chlorinated with  $CCl_4$  in a similar fashion:

 $Nb_2O_5 + 5 CCl_4 = 2 NbCl_5 + 5 COCl_2$ 265.8 1269.2 540.4 494.6

In this method (which was originally developed for analytical purposes) 1 g. of oxide and 4 ml. of  $CCl_4$  are heated for 5-10 hours at 270-300°C in a sealed tube. It is not absolutely necessary to remove the air from the tube before the reaction. After opening the tube, which should be done with the usual precautions, the reactants are distilled off. The NbCl<sub>5</sub> is resublimed under vacuum and isolated as described in method II. Because of the high pressures developed in the sealed tube, this method is limited to small quantities of reactants [E. R. Epperson et al., Inorg. Syntheses  $\underline{7}$ , 163 (1963)].

Alternate methods:

IV.

 $Nb_{9}O_{5} + 5C + 10Cl_{2} = 2NbCl_{5} + 5COCl_{2}$ 

In this very old method, the oxide is mixed with purified sugar charcoal in a 1:4 molar ratio. The granular mixture is placed (without a boat) in a tube of high-melting (Pyrex or Vycor) glass. Beyond the mixture (in the direction of gas flow) there is a fairly long bed of pure charcoal. Before starting the chlorination, both layers are dehydrated by heating to 500°C in a stream of very pure N<sub>2</sub>. Very pure, O<sub>2</sub>-free Cl<sub>2</sub> is then passed through while heating the mixture to 280-350°C and the adjacent charcoal layer to 750°C. The NbCl<sub>5</sub> receiver is sealed directly onto the reactor via a constriction. During the reaction (3 hours for 8 g. of Nb<sub>2</sub>O<sub>5</sub> and 32 g. of C) the constriction must be checked to make sure that it does not become plugged with NbCl<sub>5</sub>. The formation of the byproduct NbOCl<sub>3</sub>, which is usually difficult to avoid in this reaction, is almost completely prevented under these conditions. In spite of this, separation of the NbCl<sub>5</sub> from the NbOCl<sub>3</sub> by careful fractional sublimation is recommended. [P. Süe, Bull. Soc. Chim. France [5] <u>6</u>, 830 (1939); R. F. Rolsten, J. Amer. Chem. Soc. <u>80</u>, 2952 (1958)].

$$Nb_2O_5 + 5 CCl_4 \xrightarrow{(Cl_2)} 2 NbCl_5 + 5 COCl_2$$

A chlorine stream containing  $CCl_4$  vapor (the stream is saturated by bubbling through a  $CCl_4$ -containing wash bottle) is reacted with the oxide held in a boat which is inserted into a tube of high-melting glass or, better, a quartz tube. The reaction temperature is 300-400°C. Quite often, NbOCl<sub>3</sub> is also formed as a by-product [Gmelin-Kraut, Handbuch anorg. Chem. [Handbook of Inorganic Chemistry], 7th ed., Vol. IV/1, Heidelberg, 1928, p. 236].

VI. 
$$NbS_{2} + \frac{7}{2}Cl_{2} = NbCl_{5} + S_{2}Cl_{2}$$

[O. Hönigschmid and K. Wintersberger, Z. anorg. allg. Chem. 219, 161 (1934).]

PROPERTIES:

Yellow, granular to needle-shaped crystals; dark-red when contaminated with 1 mole % of WCl<sub>6</sub>. M.p. 204.5°C, b.p. 254°C; d 2.75. The melt is orange.

Extremely sensitive to moisture, which rapidly converts it to the white NbOCl<sub>3</sub> and then to Nb<sub>2</sub>O<sub>5</sub>  $\cdot$  xH<sub>2</sub>O; hence it cannot be handled in air without marked decomposition. Reacts vigorously with water (dec.); dissolves without decomposition in ethanol, ether and, by an unknown mechanism, also in very concentrated hydrochloric and oxalic acid solutions.

I.

$$Ta + \frac{5}{2}Cl_2 = TaCl_5$$
  
181,0 56 l. 358.2

The preparation from the elements is exactly the same as in method I for NbCl<sub>5</sub>. When Ta powder is used, the reaction starts at  $170^{\circ}$ C and is complete in a few hours at  $250^{\circ}$ C.

G. BRAUER

II.

A mixture of 2.6 g. of  $Ta_2O_5$  and 5.5 ml. of  $SOCl_2$  (threefold excess) is heated for 6 hours in a bomb tube at 230-240°C. The preparation of the starting materials and the procedure are exactly the same as for NbCl<sub>5</sub>, method II. The reaction yields a liquid solution of  $TaCl_5$  in  $SOCl_2$  from which  $SOCl_2$  is removed by distillation in the apparatus shown in Fig. 309, while the  $TaCl_5$  is resublimed.

III.

 $\begin{array}{rrrr} {\rm Ta_2O_5} &+ \ 5 \, {\rm CCl_4} &= \ 2 \, {\rm TaCl_5} &+ \ 5 \, {\rm COCl_2} \\ {\rm 441.9} & {\rm 769.2} & {\rm 716.5} & {\rm 494.6} \end{array}$ 

The reaction is carried out exactly as for NbCl<sub>5</sub>, method III, the mixture being heated to 300-320 °C.

Alternate methods: a) Methods IV and V for the preparation of NbCl<sub>5</sub> can be applied to  $TaCl_5$  in exactly the same manner. Since (in contrast to NbOCl<sub>3</sub>) no tantalum oxychloride is formed, the products are fairly pure.

b) From Ta and HCl at about 400°C [R. C. Young and C. H. Brubaker, J. Amer. Chem. Soc. 74, 4967 (1952)].

c) According to Chaigneau, the reaction between  $Ta_2O_5$  and AlCl<sub>3</sub> reported by Ruff and Thomas is nearly quantitative when the reactants are used in the following proportions:

Before use, the AlCl<sub>3</sub> should be purified by vacuum sublimation. The reactants are sealed under vacuum into a tube of high-melting glass. After heating for 48 hours at 400°C, the TaCl<sub>5</sub> product can be separated from the Al<sub>2</sub>O<sub>3</sub> by vacuum sublimation at 200°C. According to Schäfer, Göser and Bayer, reaction mixtures with a different composition (2 Ta<sub>2</sub>O<sub>5</sub> + 5 AlCl<sub>3</sub>) yield AlOCl as the residue. Mixtures of Nb<sub>2</sub>O<sub>5</sub> and AlCl<sub>3</sub> usually yield only mixtures of NbCl<sub>5</sub> and NbOCl<sub>3</sub> [O. Ruff and F. Thomas, Z. anorg. allg. Chem. <u>148</u>, 1 (1925); H. Schäfer, C. Göser and L. Bayer, Z. anorg. allg. Chem. <u>263</u>, 87 (1950); M. Chaigneau, Comptes Rendus Hebd. Séances Acad. Sci. 243, 957 (1956)].

PROPERTIES:

Colorless crystalline needles; yellow when contaminated by NbCl<sub>5</sub> (even 1% NbCl<sub>5</sub> imparts a definite yellow color) or tungsten chlorides. M.p. 216.2°C, b.p. 239°C; d 3.68. Very sensitive to moisture; decomposed by  $H_2O$  and even by concentrated HCl, separating tantalic acid. Soluble in absolute ethanol.

1306

REFERENCE:

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- II. H. Hecht, G. Jander and H. Schlapmann. Z. anorg. Chem. <u>254</u>, 255 (1947); J. Wernet. Z. anorg. allg. Chem. <u>267</u>, 213 (1952); experiments carried out in the Chemical Laboratory of the University, Freiburg i. Br., 1951.
- III. O. Ruff and F. Thomas. Z. anorg. allg. Chem. <u>156</u>, 213 (1926); H. Schäfer. Z. Naturforsch. <u>3b</u>, 376 (1948); H. Schäfer and C. Pietruck. Z. anorg. allg. Chem. <u>264</u>, 1 (1951); <u>267</u>, 174 (1951).

#### Niobium Oxytrichloride

#### NbOCl<sub>3</sub>

I.

 $\begin{array}{rcl} NbCl_{5} \ + \ \frac{1}{2}O_{2} \ = \ NbOCl_{3} \ + \ Cl_{2} \\ 270.2 \ 11.2 \ l. \ 215.3 \end{array}$ 

About 2 g. of NbCl<sub>5</sub> is allowed to sublime from a side arm into a reactor tube which is approximately 20 mm. I.D. To promote good distribution of the NbCl<sub>5</sub>, the tube contains a small amount of washed and dried glass wool. A slow stream of dry  $O_2$  (1-2 liters/hour) is passed through while the tube which is heated to 150°C by means of a tubular electric furnace. About 80% of the NbCl<sub>5</sub> reacts in 2 hours. The remainder sublimes unchanged into the cold section of the tube, from which it is driven back (vacuum) and then again treated with  $O_2$ . In this manner, nearly complete conversion is achieved; the nascent NbOCl<sub>3</sub> is sublimed in an  $O_2$ stream at 200°C to that section of the tube which is kept at 100°C. It deposits there as a dense crystal rosette. The material is discharged from the tube and handled under completely anhydrous conditions.

II. Prepared in a sealed tube according to the reaction:

 $\begin{array}{rrrr} Nb_2O_5 &+ & 3 \text{ SOCl}_2 &= & 2 \text{ NbOCl}_3 &+ & 3 \text{ SO}_2 \\ 265.8 & & 356.9 & & 430.6 & & 192.2 \end{array}$ 

The reaction proceeds exactly according to the stoichiometry shown by the equation. The method used for preparing the reactants is the same as that described for the preparation of NbCl<sub>5</sub> (p. 1303). A recommended charge for a normal bomb consists of 13.3 g. of Nb<sub>2</sub>O<sub>5</sub> (1/20 mole) and 10.9 ml. of SOCl<sub>2</sub>, prestirred into a paste.

It is important that the reactants be intimately mixed before they are heated, because otherwise the  $SOCl_2$  will react preferentially with the outer part of the oxide mass to form NbCl<sub>5</sub>, while a large portion of the oxides will remain unreacted. The mixture is heated for about 6 hours at 200°C. After cooling, well-formed crystals of NbOCl<sub>3</sub> (fine needles) are found in the lower part of the tube. Purification (by sublimation) and isolation of the NbOCl<sub>3</sub> must be done carefully: temperatures should be held below 350°C to prevent decomposition into Nb<sub>2</sub>O<sub>5</sub> and NbCl<sub>5</sub>. This operation is best carried out in the apparatus of Fig. 309.

III.

$$\frac{\text{Nb}_{2}\text{O}_{5}}{265.8} + \frac{3 \text{ Nb}\text{Cl}_{5}}{810.6} = \frac{5 \text{ Nb}\text{OCl}_{3}}{1076.5}$$

A glass tube is filled under vacuum with 0.3 g. of Nb<sub>2</sub>O<sub>5</sub> and 3 g. of NbCl<sub>5</sub> (a very large excess), with the two compounds placed at opposite ends of the tube. The inclined tube is then heated in a temperature gradient (Nb<sub>2</sub>O<sub>5</sub> 350°C/liquid NbCl<sub>5</sub> 210°C). After 12 hours, white crystalline needles of NbOCl<sub>3</sub> deposit in the center of the tube. Unreacted NbCl<sub>5</sub> is then removed by heating the tube in a 200/20°C temperature gradient. The NbOCl<sub>3</sub> is isolated under a blanket of dry, inert gas.

Alternate methods: a) The NbOCl<sub>3</sub> is often a by-product of preparative reactions for NbCl<sub>5</sub> [e.g., Nb<sub>2</sub>O<sub>5</sub> + CCl<sub>4</sub>, D. E. Sands, A. Zalkin and R. F. Elson, Acta Crystallogr. <u>12</u>, 21 (1959)]. The separation from NbCl<sub>5</sub> can be achieved by repeated fractional sublimation under vacuum or in an O<sub>2</sub> stream at temperatures below 350°C.

b) The product can also be prepared by thermal decomposition of NbCl<sub>5</sub> etherate at 90°C. [F. Fairbrother, A. H. Cowley and N. Scott (1959)].

PROPERTIES:

Colorless, often crystallizes in very fine needles. Vapor pressure: 10 mm. (234°C); 760 mm. (335°C). Disproportionates into Nb<sub>2</sub>O<sub>5</sub> (or Nb<sub>3</sub>O<sub>7</sub>Cl) and NbCl<sub>5</sub> above 350°C. Best purified by vacuum sublimation at 200°C. Very sensitive to moisture; decomposed by  $H_2O$ . Tetragonal crystals.

**REFERENCE:** 

- I. P. Süe. Bull. Soc. Chim. France [5] <u>6</u>, 830 (1939); F. Fairbrother, A. H. Cowley and N. Scott. J. Less-Common Metals <u>1</u>, 206 (1959).
- II. H. Hecht, G. Jander and H. Schlapmann. Z. anorg. Chem. <u>254</u>, 260 (1947); J. Wernet. Z. anorg. allg. Chem. <u>276</u>, 213 (1952).
- III. H. Schäfer and F. Kahlenberg. Z. anorg. allg. Chem. <u>305</u>, 327 (1960).

1 308

#### Niobium (III) Bromide

#### NbBr<sub>3</sub>

 $\frac{\text{NbBr}_{5}}{492.5} + \frac{\text{H}_{2}}{22.4 l.} = \frac{\text{NbBr}_{3}}{332.7} + 2 \text{HBr}_{332.7}$ 

The apparatus of Fig. 310, containing boat s with the niobium metal, is dried in a stream of very pure  $N_2$  at 200°C. Furnace  $o_1$  is then heated to 450°C and dry  $Br_2$  vapor is introduced in an  $N_2$  stream. The nascent NbBr<sub>5</sub> condenses at a.



Fig. 310. Preparation of niobium (III) bromide. s) boat containing niobium metal; b) asbestos wool;  $o_1, o_2$ ) electric furnaces.

After complete bromination, the bromide is sublimed in a pure  $N_2$  stream at 270°C, passing through glass wool plug *b* and the constriction *c* into section *d*. Then, a stream of high-purity  $H_2$  is introduced and NbBr<sub>5</sub> is allowed to sublime slowly into tube section *e*, kept at 500° by means of furnace  $o_2$ . It deposits on the tube wall as a shiny black crust and as a black cone on the cold finger. The tube is opened and the product is removed under a protective  $N_2$  blanket. The outer crusts are not air-sensitive and are insoluble in  $H_2O$ . On the other hand, the cones deposited in the inner part of the apparatus decompose rapidly in the presence of moist air.

#### PROPERTIES:

Black, with varying air sensitivity, depending on preparative conditions. Almost completely resistant to  $H_2O$  and dilute acids. Decomposed by concentrated  $H_2SO_4$  and  $HNO_3$ . Insoluble in organic solvents. Can be sublimed in a high vacuum ( $10^{-4}$  mm.) at about 400°C. Thermal decomposition into NbBr<sub>5</sub> and Nb begins at 900°C.

The TaBr<sub>3</sub> and the TaBr<sub>2</sub> can be obtained in the same manner as NbBr<sub>3</sub>, starting from TaBr<sub>5</sub> and H<sub>2</sub> at 700°C; however, the purity and the yield are lower [R. C. Young and T. J. Hastings, J. Amer. Chem. Soc. <u>64</u>, 1740 (1942)].

The NbBr<sub>2</sub> can be prepared from NbBr<sub>5</sub> and activated H<sub>2</sub> at 200°C; see preparation of TaBr<sub>4</sub>, p. 1310 [V. Gutmann and H. Tannenberger, Monatsh. Chem. <u>87</u>, 769 (1956)].

REFERENCE:

G. H. Brubaker and R. C. Young. J. Amer. Chem. Soc. <u>73</u>, 4179 (1951).

## Tantalum (IV) Bromide

#### TaBr₄

 $\begin{array}{l} TaBr_5 + H = TaBr_4 + HBr\\ 580.5 & 500.6 \end{array}$ 

The horizontal reaction tube of high-melting glass (about 3 cm. I.D. and 50 cm. long) shown in Fig. 311 is heated in vacuum, and a boat containing  $TaBr_5$  is introduced under anhydrous conditions. High-purity, dry H<sub>2</sub> is passed through the tube at a rate of about one liter per hour.



Fig. 311. Preparation of tantalum (IV) bromide. a) boat containing TaBr<sub>5</sub>; f) cold trap; h) grooved stopcock for fine flow regulation; s) induction coil; v) vacuum.

A pressure of 4-6 mm. is maintained in the tube by means of the grooved stopcock h, which regulates the vacuum v. A glow discharge is produced by coil s, which consists of 16 turns of copper wire (2 mm. O.D.) on the outside of the tube; a highfrequency current of 4000 kilocycles/second is applied to the coil, which consumes about 20 watts. The glass wall temperature at the coil is about 180-200°C. The tube section containing TaBr<sub>5</sub> is heated; the TaBr<sub>5</sub> is slowly vaporized and reacts with the H<sub>2</sub> activated in the glow discharge zone. The nascent TaBr<sub>4</sub> is deposited on the tube wall in this zone. Unreacted TaBr<sub>5</sub>, which condenses on the cooler portions of the tube, can be driven back into the reaction zone and reduced by moving the induction coil to another spot. The reaction of 0.8 g. of TaBr<sub>5</sub> is complete within 3 hours. The tube is allowed to cool and the product is scraped off the tube wall under anhydrous conditions. PROPERTIES:

Black powder with steel-blue tinge. Slightly hygroscopic. Disproportionates to  $TaBr_5$  and  $TaBr_3$  at 300°C under vacuum. Yields a brown solution and an insoluble residue with  $H_2O$ .

**REFERENCE:** 

V. Gutmann and H. Tannenberger. Monatsh. Chem. 87, 769 (1956).

# Niobium (V) and Tantalum (V) Bromides NbBr<sub>5</sub>, TaBr<sub>5</sub>

I.

 $\frac{Nb}{92.9} + \frac{5}{2}Br_2 = \frac{NbBr_5;}{92.5} + \frac{Ta}{181.0} + \frac{5}{2}Br_2 = TaBr_5$ 

Pure, dry N<sub>2</sub> is saturated with Br<sub>2</sub> in a washing bottle or a saturation tube s (Fig. 312); the gas mixture is then passed over Nb or Ta metal contained in the horizontal reactor tube r made of quartz or high-melting glass. The metal may be either powder or solid.



Fig. 312. Preparation of niobium (V) and tantalum (V) bromides. a) receiver; f) cold finger; m)boat containing the metal; o) tubular electric furnace; r) reactor tube; s) saturation tube.

First, the air is completely removed from the reactor. Then, the section of the tube containing the metal is heated by a coil wound directly on the reactor, or by a tubular electric furnace o. If Nb powder is used, then bromide formation begins at 90°C; with solid Nb, it starts at 195°C and with Ta powder at 155°C; it is complete in a few hours at 230-250°C. The nascent bromide sublimes onto cold finger f. When larger quantities are desired, large-diameter receiver a is attached to the end of the reactor. The extremely hygroscopic bromide should be removed from the apparatus while the latter is in a dry box (Part I, p. 71).

II.  $3 \operatorname{Nb}_2O_5(3 \operatorname{Ta}_2O_5) + 10 \operatorname{AlBr}_3 = 6 \operatorname{NbBr}_5(6 \operatorname{TaBr}_5) + 5 \operatorname{Al}_2O_3$ 797.5 1325.7 2667.3 2954.9 3483.2 509.8

In the method of Chaigneau, the mixture of pentoxide and AlBr<sub>3</sub>, in proportions indicated by the above equation, is sealed under vacuum into a Pyrex glass tube (before use, the AlBr<sub>3</sub> is purified by vacuum sublimation). The tube is heated for 24 hours at 200°C, and allowed to cool; the small amounts of Br<sub>2</sub> formed in the process and residual AlBr<sub>3</sub> are vacuum-sublimed at 140°C. The pure pentahalide is then separated from the Al<sub>2</sub>O<sub>3</sub> by vacuum sublimation at 240°C, yielding large crystals.

III. 
$$Nb_2O_5 (Ta_2O_5) + 3 C + 6 Br_2 = 2 NbBr_5 (2 TaBr_5) + COBr_2 + 2 CO_2$$
  
265.8 441.9 36.0 959.0 985.0 1161.1 187.8 88.0

The bromides are prepared by a method similar to that presented for NbCl<sub>5</sub> (or TaCl<sub>5</sub>) (method IV). An intimate mixture of the pentoxide with very pure charcoal (preferably sugar charcoal) is heated in a stream of inert gas (CO<sub>2</sub>, N<sub>2</sub>) carefully prepurified to remove traces of O<sub>2</sub> and H<sub>2</sub>O and saturated with dry Br<sub>2</sub> in a wash bottle. There is a possibility of a side reaction leading to the oxybromide in the case of Nb<sub>2</sub>O<sub>5</sub>; however, this does not happen with Ta<sub>2</sub>O<sub>5</sub>. Wiseman and Gregory report a reaction temperature of 700-860°C in the case of Ta<sub>2</sub>O<sub>5</sub>. The final product is resublimed under an inert gas or, better, in a high vacuum (190-200°C); because of its high sensitivity to moisture, it should be handled only under anhydrous conditions.

Alternate methods: a) Reaction of the pentoxide with  $CBr_4$ (analogous to that with  $CCl_4$ , see p. 1306) in a sealed tube yields pure TaBr<sub>5</sub> (in the case of Ta<sub>2</sub>O<sub>5</sub>) according to the equation: Ta<sub>2</sub>O<sub>5</sub> + 5 CBr<sub>4</sub> = 2 TaBr<sub>5</sub> + 5 CO + 5 Br<sub>2</sub>. The reactant mixture is heated for 7 days at 200°C, the gases formed are allowed to escape and TaBr<sub>5</sub> is vacuum-sublimed at 300°C. The yield is only about 70%. The corresponding reaction with Nb<sub>2</sub>O<sub>5</sub> does not yield pure NbBr<sub>5</sub>; instead, mixtures are formed [M. Chaigneau, Comptes Rendus Hebd. Séances Acad. Sci. <u>248</u>, 3173 (1959)].

b) The TaBr<sub>5</sub> can also be obtained from Ta and HBr at  $375^{\circ}$ C [R. C. Young and C. H. Brubaker, J. Amer. Chem. Soc. <u>74</u>, 4967 (1952)].

PROPERTIES:

NbBr<sub>5</sub>: red crystals; m.p. 265.2°C, b.p. 361.6°C.

TaBr<sub>5</sub>: yellow crystals; m.p. 265.8°C, b.p. 348.8°C. d 5.0.

Both compounds are very sensitive to hydrolysis, very soluble in ethanol (with a chemical reaction), and somewhat soluble in  $CCl_4$ .

REFERENCES:

- I. M. Alexander and F. Fairbrother. J. Chem. Soc. (London) <u>1949</u>, 223; D. H. Nowicky and I. E. Campbell in: H. S. Booth, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 130; R. F. Rolsten. J. Phys. Chem. <u>62</u>, 126 (1958); K. R. Krishnaswami. J. Chem. Soc. (London) <u>1930</u>, 1277; C. H. Brubaker and R. C. Young. J. Amer. Chem. Soc. <u>73</u>, 4179 (1951); W. Littke. Thesis, Univ. of Freiburg i. Br., 1961.
- II. M. Chaigneau. Comptes Rendus Hebd. Séances Acad. Sci. <u>243</u>, 957 (1956).
- III. W. K. van Haagen. J. Amer. Chem. Soc. <u>32</u>, 729 (1910); W. H. Chapin and E. F. Smith. J. Amer. Chem. Soc. <u>33</u>, 1499 (1911);
   E. L. Wiseman and N. W. Gregory. J. Amer. Chem. Soc. <u>71</u>, 2344 (1949).

## Niobium Oxytribromide

#### NbOBr<sub>3</sub>

I.

 $\frac{Nb_2O_5}{265.8} + 3 CBr_4 = 2 NbOBr_3 + 3 CO + 3 Br_2}{995.0} + \frac{3}{697.3} + \frac{3}{67.2} l. + \frac{3}{67.2} l.$ 

A stoichiometric mixture of  $Nb_2O_5$  and  $CBr_4$  is heated for 24 hours at 200°C in an evacuated, sealed tube. The tube is opened at its thin, drawn-out end, and the gases present are allowed to escape. The NbOBr<sub>3</sub> is then purified by vacuum-sublimation at 300°C. The yield is nearly quantitative.

II. NbBr<sub>5</sub> +  $\frac{1}{2}O_2$  = NbOBr<sub>3</sub> + Br<sub>2</sub> 492.5 11.2 l. 348.7

The apparatus is similar to that used for preparation of NbOCl<sub>3</sub> (method I); the NbBr<sub>5</sub> is heated in an  $O_2$  stream at 150°C. About 1 hour is necessary for 1 g. of NbBr<sub>5</sub>. The NbOBr<sub>3</sub> product is then vacuum-sublimed at 180°C into another section of the reactor and kept there at 90°C. A dense crystal deposit is obtained. The final NbOBr<sub>3</sub> must be isolated and handled under completely anhydrous conditions.

Alternate methods: a) Reaction of  $Nb_2O_5$  with C and  $Br_2$  at 540°C.

b) Decomposition of NbBr<sub>5</sub> etherate at  $112^{\circ}C$  [F. Fairbrother, A. H. Cowley and N. Scott (1959)].

TaOBr<sub>3</sub> can be prepared from TaBr<sub>5</sub> and O<sub>2</sub> at 200 °C via method II, but cannot be sublimed without decomposition.

PROPERTIES:

Yellow-brown; moisture sensitive, fumes in moist air. Thermal decomposition into  $Nb_2O_5$  and  $NbBr_5$  begins above 320°C.

REFERENCES:

- I. M. Chaigneau. Comptes Rendus Hebd. Séances Acad. Sci. 248, 3173 (1959).
- II. F. Fairbrother, A. H. Cowley and N. Scott. J. Less-Common Metals <u>1</u>, 206 (1959).

# Nobium (IV), Nobium (III) and Nobium (II) Iodides

## NbI<sub>4</sub>, NbI<sub>3</sub>, NbI<sub>3</sub>

NbI4

A tube, dried by fanning with a flame and prepared for evacuation and melt-sealing, is charged with a small amount of  $NbI_5$  under completely anhydrous conditions.

The tube is then sealed in a high vacuum and the end containing NbI<sub>5</sub> is heated to 270°C while the reactor is in a horizontal position. The liberated iodine collects at the other end, which is kept at a temperature of about 35°C ( $p[I_2] = 0.8 \text{ mm.}$ ). The reaction time is about 48 hours. A residue of NbI<sub>4</sub> remains on the spot where the starting NbI<sub>5</sub> was placed; it can be sublimed at about 300°C under the above-indicated I<sub>2</sub> pressure.

 $NbI_3$ 

Either NbI<sub>5</sub> or NbI<sub>4</sub> is heated under vacuum in a horizontal sealed tube, as described in the preparation of NbI<sub>4</sub>. The higher iodide is heated to 425-430 °C and the tube end in which the liberated I<sub>2</sub> collects is kept at 40 °C. Reaction time is 48 hours. The NbI<sub>3</sub> formed can be resublimed in the tube at 450-500 °C (partial decomposition).

NbI2

$$\begin{array}{rcl} NbI_{3} &+ \frac{1}{2}H_{2} &= NbI_{2} &+ HI \\ 473.6 & 11.2 \ l. & 346.7 & 22.4 \ l. \end{array}$$

A boat containing NbI<sub>3</sub> is heated in a stream of pure  $H_2$ . The reaction begins at 300°C and is complete in a few hours at 400°C. Higher temperatures should be avoided to prevent reduction to Nb metal or Nb hydride (these reactions start above 400°C).

PROPERTIES:

NbI<sub>4</sub>: dark-gray crystalline oblong leaflets or thin needles; metallic luster. Soluble in  $H_2O$  and dilute hydrochloric acid.

NbI<sub>3</sub>: insoluble in H<sub>2</sub>O or conc. HCl.

NbI<sub>2</sub>: gray-black. Insoluble in organic solvents; slowly hydrolyzed by  $H_2O$ . d 5.18.

**REFERENCE:** 

M. Chaigneau. Comptes Rendus Hebd. Séances Acad. Sci. <u>242</u>, 263 (1956); <u>245</u>, 1805 (1957); J. D. Corbett and P. X. Seabaugh. J. Inorg. Nuclear Chem. <u>6</u>, 207 (1958).

## Niobium (V) Iodide

NbI5

 $\begin{array}{rcl} Nb &+& {}^{5}\!/_{2}\,I_{2} &=& NbI_{3} \\ 92.9 && 634.6 && 737.5 \end{array}$ 

A vertical tube of Vycor or Pyrex glass (I.D. approximately 23 mm., wall thickness 2.5 mm.) is charged with 4-12 g. of Nb metal (either solid or powder). A dense glass wool plug is placed over the charge and approximately 20 cm. from the closed tube end, followed by a 20% excess of pure, resublimed  $I_2$  powder. The tube and its contents are thoroughly degassed in a high vacuum and melt-sealed under vacuum. Then the reactor is placed in a slightly inclined position (with the Nb metal at the higher and the  $I_2$  at the lower end) and heated by means of two separate tubular electric furnaces (these meet at the center of the tube). The niobium is heated to 300°C and the  $I_2$  first to 180°C and then to 250°C. The reaction is nearly quantitative after 10-15 hours and NbI<sub>5</sub> crystals collect in the transition zone between the two temperature regions. The yields are lower with Nb powder than with solid Nb. The reactor is broken at the center, the NbI<sub>5</sub> is

removed under anhydrous conditions (e.g., in a drybox, see Part I, p. 71), and repeatedly rinsed with dry petroleum ether (under  $N_2$ ) until the adhering  $I_2$  is removed and the petroleum ether stays colorless. The traces of petroleum ether are evaporated in vacuum.

According to Corbett and Seabaugh, this synthesis can also be carried out in a V-shaped, closed reactor tube.

The method of reacting a pentoxide with  $AlI_3$ , used successfully for the preparation of  $TaI_5$ , yields only impure  $NbI_5$  when  $Nb_2O_5$ is the starting material [M. Chaigneau, Comptes Rendus Hebd. Séances Acad. Sci. <u>242</u>, 263 (1956)].

Alternate method: Repeated distillation of NbBr<sub>5</sub> in an HI stream (a pure product is not readily obtained, however) [W. M. Barr, J. Amer. Chem. Soc. <u>30</u>, 1568 (1908); W. K. van Haagen, J. Amer. Chem. Soc. <u>32</u>, 729 (1910)].

PROPERTIES:

Yellow leaflets or needle-shaped crystals with a brass luster. Sublimes without decomposition only under considerable  $I_2$  pressure. Very sensitive to moisture; decomposed by  $H_2O$ , forming HI.

**REFERENCE:** 

F. Körösy. J. Amer. Chem. Soc. <u>61</u>, 838 (1939); K. M. Alexander and F. Fairbrother. J. Chem. Soc. (London) <u>1949</u>, 2472; R. F. Rolsten. J. Amer. Chem. Soc. <u>79</u>, 5409 (1957); J. D. Corbett and P. X. Seabaugh. J. Inorg. Nuclear Chem. <u>6</u>, 207 (1958); W. Littke. Thesis, Univ. of Freiburg i. Br., 1961.

# Tantalum (V) Iodide

TaI₅

I.

$$T_a + {}^{5}/_{2}I_{2} = T_aI_{5}$$
  
180.9 634.6 815.5

The procedure corresponds exactly to that described for NbI<sub>5</sub>. The tube end containing the Ta metal is heated to 300°C and that containing the  $I_2$  first to 180, then to 250°C. The reaction is complete in 10-15 hours.

II.  $\begin{array}{rl} 3 \operatorname{Ta_2O_5} + 10 \operatorname{AlI_3} &= 6 \operatorname{TaI_5} + 5 \operatorname{Al_2O_3} \\ 1325.7 & 4077.1 & 4893.0 & 509.8 \end{array}$ 

In this method of Chaigneau, a stoichiometric mixture of  $Ta_2O_5$ and All<sub>3</sub>, in a Pyrex glass tube, is heated in vacuum for 24 hours at 230°C. The tube section containing the reaction mixture (which by then is black) is heated further to 350°C and finally to 520°C. The TaI<sub>5</sub> sublimes (nearly theoretical yield) into the colder end of the tube, where it deposits as crystals.

PROPERTIES:

Shiny black rhombic crystals, subliming at 543°C. Vapor pressure: 7.6 mm. (320°C); 96 mm. (420°C); 421 mm. (500°C). d 5.80. Very sensitive to moisture.

REFERENCE:

- I. F. Körösy. J. Amer. Chem. Soc. <u>61</u>, 838 (1939); K. M. Alexander and F. Fairbrother. J. Chem. Soc. (London) <u>1949</u>, 2472;
   R. F. Rolsten. J. Amer. Chem. Soc. <u>80</u>, 2952 (1958); W. Littke. Thesis, Univ. of Freiburg i. Br., 1961.
- II. M. Chaigneau. Comptes Rendus Hebd. Séances Acad. Sci. <u>242</u>, 263 (1956).

## Niobium (II) Oxide

#### NbO

I.

 $NbO_2 + Nb = 2 NbO$ 124.9 92.9 217.8

A mixture of NbO<sub>2</sub> and Nb metal is pulverized to as small a size as possible and then compressed into small pellets which are heated for 10-20 minutes at 1600-1700°C in an atmosphere of very pure Ar or in a high vacuum. The best support for these pellets is Nb sheet; however, the pellets should touch this sheet only at a very few points.

II. NbO can also be prepared via a prolonged reduction of higher niobium oxides with  $H_2$ . An especially thorough prepurification and predrying of the hydrogen is essential. The reaction is carried out at 1300-1750°C. The reduction of 0.5 g. of NbO<sub>2</sub> to NbO takes about 60 hours at 1350° and about 15 hours at 1700°C. It is important to follow the progress of the reduction via a periodic check on the weight of the reactants; this is because the reaction also readily yields Nb metal in addition to NbO. The metal may start to accumulate after the run is in progress for some time and, in the presence of the unavoidable trace impurities in  $H_2$ , may be converted into Nb<sub>2</sub>N and Nb<sub>2</sub>C.

PROPERTIES:

Formula weight 108.91. Gray, submetallic. d 7.30. Crystal structure: special type similar to B 1 (NaCl) type.

**REFERENCE:** 

G. Grube, O. Kubaschewski and K. Zwiauer. Z. Elektrochem. <u>45</u>, 885 (1939); O. Kubaschewski. Z. Elektrochem. <u>46</u>, 284 (1940);
G. Brauer. Z. anorg. allg. Chem. <u>248</u>, 1 (1941).

## Niobium (IV) Oxide

#### NbO<sub>2</sub>

In this procedure, pure  $Nb_2O_5$  is reduced in an H<sub>2</sub> stream at 1000-1200°C.

$$\frac{Nb_2O_5}{265.8} + \frac{H_2}{22.4} = 2 \frac{NbO_2}{249.8} + \frac{H_2O}{18.0}$$

The reduction time for 1 g. of oxide (contained in a boat) is 1-2 hours. Weight control is necessary, since prolonged heating at high temperatures produces some further reduction to NbO.

PROPERTIES:

Formula weight 124.91. Black powder. d 5.9. Crystal structure: C 4 (rutile) type.

REFERENCE:

 P. Klinger. Techn. Mitteil. Krupp, Forschungsber. <u>1939</u>, p. 171;
 G. Grube, O. Kubaschewski and K. Zwiauer. Z. Elektrochem. <u>45</u>, 885 (1939);
 G. Brauer. Z. anorg. allg. Chem. <u>248</u>, 1 (1941).

## Niobium (V) and Tantalum (V) Oxides

## $Nb_2O_5$ , $Ta_2O_5$

Commercial Nb<sub>2</sub>O<sub>5</sub> and Ta<sub>2</sub>O<sub>5</sub> are usually low-purity products; in particular, Nb<sub>2</sub>O<sub>5</sub> often contains Ta<sub>2</sub>O<sub>5</sub>. They are frequently contaminated with Fe, Ti and Sn since these elements accompany Nb and Ta in the original minerals. The following methods for purifying the pentoxides are based on the assumption that the content of these impurities does not exceed a few percent.

1318

## I. PURIFICATION VIA THE CHLORIDE

a) The tantalum and the tin can be removed from commercial  $Nb_2O_5$  (or  $Ta_2O_5$ ) by converting the oxide into a chloride, followed by extraction or distillation. This method is limited to small quantities of reactants.

As we have shown in the preparation of NbCl<sub>5</sub> (or TaCl<sub>5</sub>) (p. 1304) 1 g. of the oxide and 4 ml. of CCl<sub>4</sub> are placed in a common tubular bomb (which need not be evacuated), and chlorination is carried out while heating to 250-300 °C. The sealed tube is then opened, and the solid pentachloride product is extracted five times by shaking with 5-ml. portions of CCl<sub>4</sub>, followed by phase separation. This may be done in air provided the operation is carried out quickly. The TiCl<sub>4</sub> and SnCl<sub>4</sub> are very readily soluble in CCl<sub>4</sub>, whereas the pentachlorides dissolve less readily and less rapidly (less than 10 mg. of chloride/1 ml. of CCl<sub>4</sub>) and therefore remain as residues. The residues are then converted to the oxides with H<sub>2</sub>O. Assuming complete chlorination prior to the extraction, the Ti and Sn content of the product oxides should be  $\geq 0.05\%$ .

b) Instead of leaching the  $TiCl_4$  and  $SnCl_4$  out of the chlorination product, the latter can be removed by vacuum sublimation at 0.1 mm. and 200°C (after the sealed tube is opened). In this procedure the subliming pentachlorides travel only a short distance within the tube before depositing in a cooler zone; however, the  $TiCl_4$  and  $SnCl_4$  are volatilized so completely that the pentoxides obtained by this procedure contain less than 0.05% of  $TiO_3$  and  $SnO_2$ .

c) The following method, which can be used to purify  $Nb_2O_5$  (but not  $Ta_2O_5$ ), has the advantage over the previously described one that it can be used with larger quantities of reactants.

The following preliminary treatments may be used:

a)  $Nb_2O_5$  is fused with KHSO<sub>4</sub>; the melt is allowed to cool and, after the grinding, is treated with dilute  $H_2SO_4$  and  $H_2O_2$ . The hydrated oxide is precipitated from the peroxide solution with  $SO_2$  at the boiling point, the mixture is decanted, the supernatant is discarded and the aqueous slurry of the precipitate is used in further reactions.

b) Freshly precipitated hydrated oxide (or the  $Nb_2O_5$ -KHSO<sub>4</sub> melt) is directly dissolved in ammonium oxalate or tartaric acid and the resulting solution of the complex is used in further reactions.

c) A hydrochloric acid solution or suspension of the chlorination products (NbCl<sub>5</sub>, NbOCl<sub>3</sub>) may be the starting material.

Next, the solution (or suspension) is adjusted to a volume corresponding to a maximum concentration of 4 g. (or, even better, 2-3 g.) of Nb<sub>2</sub>O<sub>5</sub>/100 ml. and the solution is saturated, while ice-cooling and agitating, with HCl gas as described in detail in the preparation of  $(NH_4)_2TiCl_6$ , p. 1199 ff. The suspension, as well as

the hydrated oxide precipitated during the treatment with HCl, eventually becomes clear when the solution is saturated with HCl; sometimes, however, the HCl treatment and the agitation must be quite long. Approximately three hours are required per 100 Then 4 g. of solid NH<sub>4</sub>Cl per 100 ml. is added and the ml. mixture is agitated for approximately half an hour. The (NH<sub>4</sub>)<sub>2</sub>TiCl<sub>6</sub> and (NH<sub>4</sub>)<sub>2</sub>SnCl<sub>6</sub> precipitate up to their respective solubility limits (0.5 mg. of Ti and 0.4 mg. of Sn/100 ml.). The mixture is filtered through a small-pore fritted-glass filter (it is best to cool the filter externally with ice); the filtrate is diluted with four to five times its volume of H<sub>2</sub>O and then hydrolyzed at the boiling point. The readily filtered hydrated oxide is then calcined to the oxide. A 70-80% yield of purified Nb<sub>2</sub>O<sub>5</sub>, containing less than 0.1% of TiO<sub>2</sub> or SnO<sub>2</sub>, is obtained. The losses are due to the isomorphous occlusion of  $(NH_4)_2NbOCl_5$  by the precipitated  $(NH_4)_2 TiCl_4$ .

#### **II.** SEPARATION VIA THE OXALATE

Very pure niobium oxide can be obtained either from crude niobium oxide or from concentrated niobium oxide mixtures, provided the  $Ta_2O_5$ :Nb<sub>2</sub>O<sub>5</sub> ratio is not greater than 1:4. The following method is used: Precipitated, moist hydrated oxide (equivalent to about 20 g. of anhydrous oxide) is repeatedly treated with fresh 200-ml. portions of a solution which is 2N in HCl and 5% in oxalic acid dihydrate. Each treatment involves heating the mixture for several hours on a bath at 60-70°C with stirring, followed by decantation. Most of the Ta remains in the insoluble residue, which also becomes concentrated in Ti and W; the nearly pure Nb goes into solution, always accompanied by Sn. The combined solutions are evaporated to dryness, and the residue calcined to decompose the oxalic acid. The calcination residue is rather pure Nb<sub>2</sub>O<sub>5</sub>, containing only about 1% Ta<sub>2</sub>O<sub>5</sub> plus some alkali stemming from the starting material.

This fairly pure  $Nb_2O_5$ , in the form of precipitated hydrated oxide, may again be subjected to the same leaching operation, which gives a virtually Ta-free oxide.

## **III.** SEPARATION BY EXTRACTION

The extraction of aqueous hydrofluoric solutions of the pentoxides with immiscible ketones can be used with very impure starting materials. In this case there is no restriction on the permissible Nb: Ta ratio. In this extraction, the aqueous phase contains HF and either HC1,  $HNO_3$  or  $H_2SO_4$ ; the best purification efficiency is obtained with  $HNO_3$  and  $H_2SO_4$ . The coefficients for partition of Nb and Ta between the two phases depend to a large extent on the acid concentration. The extracting agents may be methyl isobutyl ketone (MIBK) or cyclohexanone. All equipment must be made of an HF-resistant material (polyethylene or polyvinyl chloride).

The pentoxide (or a mixture of pentoxides) is dissolved in strong hydrofluoric acid and the resulting solution adjusted, by addition of  $H_2SO_4$ , to a concentration of 100 g. of pentoxide/liter, 5.6 N HF and 9 N  $H_2SO_4$ . This solution is extracted twice with half its volume of MIBK. The combined ketone extracts contain virtually all of the Nb and Ta originally present and are free of other metals. The organic phase is then extracted with aqueous acid as follows.

a) When the starting material is either  $Nb_2O_5$  or an Nb-rich mixture, the organic phase is shaken with the same volume of an aqueous solution that is 3 N in  $H_2SO_4$  and 1 N in HF. This aqueous phase then contains 90% of the Nb, but less than 0.1% of Ta. A second extraction of the ketone phase with fresh acid solution gives a second, smaller Nb fraction of lower purity. A small amount of high-purity Ta remains in the organic phase.

b) When the starting material is  $Ta_2O_5$  or a Ta-rich mixture, the organic phase stemming from the first extraction is shaken with the same volume of an aqueous acid solution which is 4.5 N in  $H_2SO_4$  and 2.8 N in HF, thus removing all of the Nb together with a small amount of Ta. The remaining ketone solution contains most of the Ta (99.9% purity).

Pure Nb or Ta is obtained from the ketone solutions by extraction with pure  $H_2O$ . The pentoxide, dissolved in the aqueous (more or less acidic) final solutions, is then precipitated with ammonia.

In special cases or when the starting materials contain a moderate Nb: Ta ratio the procedure can be modified. These modifications are summarized in the following table.

Extraction of Nb and Ta with Methyl Isobutyl Ketone (MIBK)

Extraction system	Optimum extraction for Ta alone (from the aqueous phase)	Optimum extraction for Nb or Nb + Ta (from the aqueous phase)		
$(HF + HNO_3)/MIBK$ $(HF + H_2SO_4)/MIBK$ (HF + HC1)/MIBK	0.5 N HF + 1 N HNO <sub>3</sub> 1 N HF + 3 N H <sub>2</sub> SO <sub>4</sub> 3 N HF + 3 N HC1 or 3 N HF	7 N HF + 5 N HNO $_{3}$ 5.6 N HF + 9 N H <sub>2</sub> SO <sub>4</sub> 6.5 N HF + 7.2 N HC1		
Alternate methods:

#### IV. FLUORIDE SEPARATION BY THE METHOD OF MARIGNAC

Fractional crystallization of  $K_2TaF_7$  is accomplished by adding KF to hydrofluoric solutions of Nb and Ta, while  $K_2NbOF_5$  accumulates in the mother liquor. This procedure is more suitable for a large-scale process than the laboratory [O. Hönigschmid and K. Wintersberger, Z. anorg. allg. Chem. <u>219</u>, 161 (1934); C. W. Balke, Trans. Electrochem. Soc. <u>85</u>, 89 (1944); G. S. Savchenko and Ya. V. Tananayev, Zh. Prikl. Khimii <u>19</u>, 1093 (1946); <u>20</u>, 385 (1947)].

V. TANNIN PRECIPITATION

Small amounts (about 1 g.) of very pure  $Nb_2O_5$  and  $Ta_2O_5$  can be obtained from the corresponding crude oxides via a simple fractional precipitation of oxalate complex solutions with tannin. The procedure is based on the analytical method of Schoeller (1937).

#### VI. SEPARATION WITH ION EXCHANGERS

The separation of Nb and Ta on an ion exchange column is promising but not yet sufficiently developed. The Nb and Ta products can be obtained in 99% purity from solutions which are 9 M in HCl and 0.05 M in HF [K. A. Kraus and G. D. Moore, J. Amer. Chem. Soc. <u>71</u>, 3855 (1949); E. H. Huffman, G. M. Iddings and R. C. Lilly, J. Amer. Chem. Soc. <u>73</u>, 4474 (1951); J. L. Hague, E. D. Brown and H. A. Bright, J. Res. Nat. Bur. Standards <u>53/4</u>, 261 (1954); P. Münchow, Chem. Ztg. <u>84</u>, 490, 527 (1960)].

#### PROPERTIES:

Nb<sub>2</sub>O<sub>5</sub>: White powder turning yellow on heating; insoluble in aqueous acids other than hydrofluoric. M.p. 1495°C. Crystallizes in various modifications; does not form well-defined hydrates.

 $Ta_{2}O_{5}$ : White powder; insoluble in aqueous acids other than hydrofluoric. M.p. 1872°C. Crystallizes in various modifications; forms no defined hydrates.

#### REFERENCES:

- General: W. R. Schoeller. The Analytical Chemistry of Tantalum and Niobium, London, 1937; H. Schäfer. Angew. Chem. <u>71</u>, 153 (1959); G. L. Miller. Tantalum and Niobium, London, 1959.
  - I. R. F. Weinland and L. Storz. Z. anorg. allg. Chem. <u>54</u>, 223 (1907); H. Schäfer and C. Pietruck. Z. anorg. allg. Chem. <u>264</u>, 106 (1951).

- II. H. Schäfer, L. Bayer and C. Pietruck. Z. anorg. allg. Chem. <u>266</u>, 140 (1951); J. Wernet, Z. anorg. allg. Chem. <u>267</u>, 213 (1952).
- III. K. B. Higbie and J. R. Werning. U. S. Bur. Min. Rep. Invest. No. 5239 (1956); E. L. Koerner, M. Smuts and H. A. Wilhelm. Meeting of Extractive Metallurgy Section of the A.I.Ch.E., Chicago, 1957; G. H. Faye and W. R. Inman. Canad. Dept. Min. Techn. Surv. Res. Rep. No. MD 210 (1957); J. L. Tews and S. L. May. U. S. Bur. Min. Rep. No. USBM-U-252 (1957); C. W. Carlson and R. H. Nielsen, J. Metals <u>12</u>, 472 (1960).

#### Alkali Niobates and Tantalates

Either  $Nb_2O_5$  or  $Ta_2O_5$  is heated with an alkali hydroxide or alkali carbonate. Salts of differing composition are obtained, depending on the reactant ratios, the temperature, and workup of the reaction product. Systematic investigation has shown that the system alkali oxide-pentoxide contains a wide range of compounds.

The thoroughly dried starting Nb<sub>2</sub>O<sub>5</sub> (or Ta<sub>2</sub>O<sub>5</sub>) and Li<sub>2</sub>CO<sub>3</sub>, Na<sub>2</sub>CO<sub>3</sub> or K<sub>2</sub>CO<sub>3</sub> are mixed in ratios calculated to give the desired final composition; the mixtures are heated in a crucible at a slowly increasing temperature (e.g., 100°C/hour) to insure a smooth reaction without loss of reactants. The best crucible material is Pt or (particularly for mixtures with high alkali concentrations) an 80% Au, 20% Pd alloy. The temperature is increased to just below the melting point of the mixture and kept at this point several hours; the mixture is then cooled and pulverized. The heating and pulverization are repeated an additional two times. An oxygen atmosphere is maintained over the crucible during the heating to prevent the product from splitting off oxygen.

In the preparation of the meta salts

 $\begin{array}{rrrr} \mathrm{Nb_2O_5} &+ \mathrm{Li_2CO_3}(\mathrm{Na_2CO_3}, \mathrm{K_2CO_3}) \rightarrow 2 \ \mathrm{LiNbO_3}(2 \ \mathrm{NaNbO_3}, 2 \ \mathrm{KNbO_3}) \\ \mathrm{265.8} & 73.9 & 106.0 & 138.2 & 295.7 & 327.8 & 360.0 \\ \mathrm{Ta_2O_5} &+ \mathrm{Li_2CO_3}(\mathrm{Na_2CO_3}, \mathrm{K_2CO_3}) \rightarrow 2 \ \mathrm{LiTaO_3}(2 \ \mathrm{NaTaO_3}, 2 \ \mathrm{KTaO_3}) \\ \mathrm{441.9} & 73.9 & 106.0 & 138.2 & 471.8 & 503.9 & 536.1 \\ \end{array}$ 

the amount of alkali carbonate may be somewhat greater than that indicated by the 1:1 molar ratio (however, a 2:1 ratio should not be exceeded) and the reaction mixture may be extracted with warm  $H_2O$ . The meta salts remain as residues which dissolve only with difficulty. Larger single crystals of the meta salts can be obtained from a KF or KCl melt.

For information concerning polyniobates and polytantalates, see Part III, Section 3.

PROPERTIES:

Colorless, crystalline compounds. The following phases are known:

Li<sub>2</sub>O - Nb<sub>2</sub>O<sub>5</sub>:

 $3 \text{ Li}_{2}O \cdot \text{Nb}_{2}O_{5}$  (m.p. 1408°C);  $\text{Li}_{2}O \cdot \text{Nb}_{2}O_{5}$  (m.p. 1253°C);  $\text{Li}_{2}O \cdot 4 \text{ Nb}_{2}O_{5}$  (m.p. 1231°C, incongruent);  $\text{Li}_{2}O \cdot 14 \text{ Nb}_{2}O_{5}$  (m.p. 1268°C, incongruent).

 $Na_{2}O - Nb_{2}O_{5}$ :

3 Na<sub>2</sub>O · Nb<sub>2</sub>O<sub>5</sub> (m.p. 997°C); Na<sub>2</sub>O · Nb<sub>2</sub>O<sub>5</sub> (m.p. 1422°C, polymorphous); Na<sub>2</sub>O · 4 Nb<sub>2</sub>O<sub>5</sub> (m.p. 1277°C, incongruent); Na<sub>2</sub>O · 14 Nb<sub>2</sub>O<sub>5</sub> (m.p. 1309°C, incongruent).

K<sub>2</sub>O - Nb<sub>2</sub>O<sub>5</sub>:

3 K<sub>2</sub>O  $\cdot$  Nb<sub>2</sub>O<sub>5</sub> (m.p. 950°C); K<sub>2</sub>O  $\cdot$  Nb<sub>2</sub>O<sub>5</sub> (m.p. 1039°C, incongruent); 2 K<sub>2</sub>O  $\cdot$  3 Nb<sub>2</sub>O<sub>5</sub> (m.p. 1163°C); K<sub>2</sub>O  $\cdot$  3 Nb<sub>2</sub>O<sub>5</sub> (m.p. 1234°C, incongruent); 3 K<sub>2</sub>O  $\cdot$  22 Nb<sub>2</sub>O<sub>5</sub> (m.p. 1279°C, incongruent); 6 K<sub>2</sub>O  $\cdot$  7 Nb<sub>2</sub>O<sub>5</sub> and 7 K<sub>2</sub>O  $\cdot$  6 Nb<sub>2</sub>O<sub>5</sub> (metastable, obtainable only by quenching).

K<sub>2</sub>O-Ta<sub>2</sub>O<sub>5</sub>:

 $3 K_2O \cdot Ta_2O_5$  (m.p. 1330°C);  $K_2O \cdot Ta_2O_5$  (m.p. 1370°C, incongruent);  $K_2O \cdot 2 Ta_2O_5$  (m.p. 1520°C, incongruent);  $K_2O \cdot 5 Ta_2O_5$  (m.p. 1645°C, incongruent).

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L. L. Quill. Z. anorg. allg. Chem. <u>208</u>, 257 (1932); P. Süe. Comptes Rendus Hebd. Séances Acad. Sci. <u>198</u>, 1696(1934); P. Süe. Ann. Chimie [11] <u>7</u>, 493 (1937); F. Windmaisser. Österr. Chemiker-Ztg. <u>45</u>, 201 (1942); B. T. Matthias and J. P. Remelka. Phys. Rev. (2) <u>82</u>, 727 (1951); E. A. Wood, Acta Crystallogr. <u>4</u>, 353 (1951); A. Reisman, F. Holtzberg, M. Berkenblit, M. Berry and E. Banks. J. Amer. Chem. Soc. <u>77</u>, 2115 (1955); <u>78</u>, 719, 4514 (1956); <u>80</u>, 37, 6503 (1958); <u>81</u>, 1292 (1959).

## Peroxyniobic and Peroxytantalic Acids HNbO4 • n H2O, HTaO4 • n H2O

#### HNbO<sub>4</sub> · n H<sub>2</sub>O

Sulfuric acid (2 N, 10 ml.) is gradually added to a solution of 2 g. of potassium peroxyniobate in 50 ml. of  $H_2O$ . The precipitate

formed (caution: this precipitate redissolves readily if there is an excess of sulfuric acid) is filtered off and washed three to four times with  $H_2O$ , ethanol and ether. The yield is poor (about 1 g. of peroxyniobic acid, in the form of a light yellow powder).

## HTaO<sub>4</sub> · n H<sub>2</sub>O

I.  $2 K_3 TaO_8 + 3 H_2SO_4 + 4 H_2O = 2 HTaO_4 + 3 K_2SO_4 + 6 H_2O_2$ 852.5 294.2 491.8 522.8 204.1

A solution of 2 g. of  $K_3 TaO_8$  in 150 ml. of  $H_2O$  is prepared and an approximately equivalent quantity of dilute  $H_2SO_4$  added all at once. The nascent precipitate is first centrifuged; it can then be filtered and washed with some  $H_2O$ , then with ethanol and ether.

II. 
$$Ta_2O_5 + 2H_2O_2 = 2HTaO_4 - H_2O_4 + H_2$$

Tantalic acid, freshly precipitated with ammonia from a solution containing 10 g. of  $K_2TaF_7$ , is treated with 50 ml. of  $H_3O$  and 50 ml. of 30%  $H_2O_2$ . The tantalic acid dissolves completely within a few hours, affording a transparent, opalescent liquid. The latter is treated with equal amounts of ethanol and saturated NaCl solution. The mixture, which at first stays clear, slowly deposits out a precipitate of peroxytantalic acid, which is filtered off the next day and dried with ethanol and ether.

PROPERTIES:

White substance. Gelantinous when wet, fine powder when dry. Ta: peroxy oxygen ratio = 1:1.

REFERENCES:

A. Sieverts and E. L. Müller. Z. anorg. allg. Chem. <u>173</u>, 297 (1928); P. Melikow and L. Pissarjewsky. Z. anorg. Chem. <u>20</u>, 344 (1899).

Potassium Peroxyniobate, Potassium Peroxytantalate

#### K<sub>3</sub>NbO<sub>8</sub>, K<sub>3</sub>TaO<sub>8</sub>

 $\begin{array}{rl} Nb_2O_5 \ + \ 6\ KOH \ + \ 8\ H_2O_2 \ = \ 2\ K_3NbO_8 \ + \ 11\ H_2O_2\\ 265.8 \ 336.6 \ 272.1 \ 676.4 \ 198.2\\ Ta_2O_5 \ + \ 6\ KOH \ + \ 8\ H_2O_2 \ = \ 2\ K_3TaO_8 \ + \ 11\ H_2O_2\\ 441.9 \ 336.6 \ 272.1 \ 852.5 \ 198.2 \end{array}$ 

 $K_3NbO_8 \cdot 1/_2 H_2O$ 

I. A mixture of  $Nb_2O_5$  (1 part) and KOH (8 parts) is fused in a silver crucible. The fused mass is dissolved in a minimum of

 $H_2O_3$  a small amount of  $H_2O_2$  is added and the mixture heated for a short while on the water bath. The solution is filtered to remove the black Ag particles, 9-10 moles of  $H_2O_2$ /mole of Nb<sub>2</sub>O<sub>5</sub> is added, and the mixture is precipitated with an equal volume of ethanol. The precipitate is air-dried and washed with ethanol and ether; it is then redissolved in a mixture of three to four moles of  $H_2O_2$  and 0.5 moles of KOH/mole of the Nb<sub>2</sub>O<sub>5</sub> reactant and precipitated with 1-1.5 times its volume of ethanol. The precipitate is again dried with ethanol and ether.

II. The fusion step can be avoided by using freshly precipitated niobic acid or potassium niobate instead of  $Nb_2O_5$ . These compounds are dissolved in potassium hydroxide and the workup procedure is the same as that described in method I.

K<sub>3</sub>TaO<sub>8</sub>

Tantalic acid  $Ta_2O_5 \cdot aq$ . is precipitated with ammonia from a tantalum solution (e.g., that of  $K_2TaF_7$ ). The precipitate is suction-filtered and washed with a large quantity of  $H_2O$ . The gelatinous intermediate, which should not be allowed to age, is added to a solution of 20 g. of very pure KOH in 250 ml. of 3%  $H_2O_2$  (made from "Perhydrol," Merck) until the solution is saturated. Cooling to 0°C causes separation of granular peroxytantalate crystals, which are suction-filtered and dried with ethanol and ether or in a vacuum desiccator over  $H_2SO_4$ .

The pure white crystals have a composition corresponding to  $K_3 TaO_8$ ; the yield is poor.

 $K_3TaO_8 \cdot 1/_2 H_2O$ 

A mixture of  $Ta_2O_5$  (5 g.) with three times the stoichiometric quantity of KOH is fused in a silver crucible. The product is allowed to cool and is then dissolved in 3%  $H_2O_2$ ; any separated silver is filtered off. The solution is treated first with 20 times the stoichiometric quantity of  $H_2O_2$  and then with an equal volume of ethanol. The precipitated, fine powder of the salt is suctionfiltered and dried with ethanol and ether. Yield: 6 g. of pure white  $K_3TaO_8 \cdot 1/2 H_2O_8$ .

REFERENCES:

A. Sieverts and E. L. Müller. Z. anorg. allg. Chem. <u>173</u>, 297 (1928); C. W. Balke. J. Amer. Chem. Soc. <u>27</u>, 1140 (1905);
C. W. Balke and E. F. Smith. J. Amer. Chem. Soc. <u>30</u>, 1637 (1908); P. Melikow and L. Pissarjewsky. Z. anorg. Chem. <u>20</u>, 344 (1899).

## **Niobium and Tantalum Sulfides**

## I. FROM THE METALS

In general, the synthesis from the elements gives products of any desired composition. If the reactant ratios correspond exactly to the region in which a phase is homogeneous, the product is pure; otherwise, it is a mixture of phases.

A mixture of about 1-3 g. of solid Nb metal (or, better, Nb filings or powder) and that quantity of vacuum-distilled sulfur which will give the desired product composition is placed in a quartz tube, which is then evacuated. The mixture is heated slowly and then kept at 700-1000°C for two days. Under these conditions, dpending on the temperature used, either the highor the low-temperature modification of a phase is obtained. Special conditions are required for some compounds (e.g., hexagonal NbS<sub>2</sub>: 850°C < T < 1050°C, with the niobium placed in the hottest zone of the ampoule; rhombohedral Nb<sub>1+x</sub>S<sub>2</sub>: T > 800°C, niobium in the coldest zone of the ampoule).

In no case is the sulfur quantity absorbed greater than that corresponding to the formula  $NbS_{2}$ . No preparative methods are known as yet for  $NbS_{3}$ . Traces of this compound are formed during the preparation of other niobium sulfides. The heating of mixtures low in sulfur should be occasionally interrupted, the intermediate product repulverized and remixed, then replaced in the evacuated tube, and the heating continued.

The lower sulfides may also be obtained by homogenization of the corresponding mixtures of niobium sulfide + Nb. In addition, partial degradation of the higher sulfides by distilling off the sulfur in a high vacuum also yields lower sulfides. Another preparative method is based on the reaction of Nb with  $H_2S$  at temperatures between 550 and 900°C.

The tantalum sulfides are prepared by procedures based on the same principles. Since the phase relationships are less complicated, one has greater lattitude in selecting the preparative conditions. The tantalum sulfide with the highest known S concentration is  $TaS_3$ , which can be obtained at 600°C.

#### **II**. FROM THE PENTOXIDES

Either Nb<sub>2</sub>O<sub>5</sub> or Ta<sub>2</sub>O<sub>5</sub> (2-10 g.) is placed in a loose layer in a porcelain reactor tube and exposed for three to six hours at 960-1300°C to a CS<sub>2</sub>-saturated stream of H<sub>2</sub>S; prior to entering the tube, the latter passes through a purification and drying train, as well as through a wash bottle containing CS<sub>2</sub> at 25-35°C. The reactor must be thoroughly purged of air prior to the run and air leaks must be avoided during the reaction. In particular, the  $H_2S$  must be air-free. The crude sulfides are then extracted with  $CS_2$  to remove traces of precipitated S. The Nb product has the composition NbS<sub>1.74</sub>, the Ta product TaS<sub>2.0</sub>. It appears that no other compositions can be obtained by this method.

PROPERTIES:

The hexagonal NbS<sub>2</sub> and the rhombohedral Nb<sub>1+x</sub>S<sub>2</sub> are blueblack, shiny crystalline compounds: NbS<sub>2</sub> single crystals 0.5 mm. in size can be obtained. The remaining niobium sulfides are dark-gray to black or dark-brown powders devoid of luster.

The TaS<sub>2</sub> consists of microscopically small leaflets, not appreciably volatile in vacuum up to 1100°C, whereas TaS<sub>3</sub> consists of a mass of loose, feltlike crystalline fibers which are always obtained when other modifications are heated for 14 days at 600°C. At 650°C, TaS<sub>3</sub> decomposes rapidly into TaS<sub>2</sub> and sulfur, which dissolves in the TaS<sub>2</sub>. The sulfides are unaffected by hydrochloric acid and sodium hydroxide, but vigorously oxidized by hot concentrated H<sub>2</sub>SO<sub>4</sub> or HNO<sub>3</sub>.

Seven phases are known in the Nb-S system: "H-NbS" (temporary general designation, probably comprising two or more phases), hexagonal Nb<sub>2-y</sub>S<sub>2</sub> (0 < y < 0.3) and the closely related hexagonal Nb<sub>1</sub> S<sub>2</sub> (0.3 < x < 0.43), rhombohedral Nb<sub>1+x</sub>S<sub>2</sub> (0.12 < x < 0.5), hexagonal NbS<sub>2</sub>, rhombohedral NbS<sub>3</sub> and monoclinic NbS<sub>3</sub>.

**REFERENCES**:

- Nb-S: W. Biltz and A. Köcher. Z. anorg. allg. Chem. <u>237</u>, 369 (1938); O. Hönigschmid and K. Wintersberger. Ibid. <u>219</u>, 161 (1934); H. Biltz and W. Gonder. Ber. dtsch. chem. Ges. <u>40</u>, 4963 (1907); G. Hägg and N. Schönberg. Ark. Kemi. <u>7</u>, 371 (1954); Hörst Müller. Thesis, Univ. of Freiburg i Br., 1938; F. Jellinek, G. Brauer and H. Müller. Nature <u>185</u>, 376 (1960).
- Ta-S: W. Biltz and A. Köcher. Z. anorg. allg. Chem. <u>238</u>, 81 (1938); H. Biltz and C. Kircher. Ber. dtsch. chem. Ges. <u>43</u>, 1636 (1910); G. Hägg and N. Schönberg. Ark. Kemi <u>7</u>, 371 (1954).

## Niobium and Tantalum Nitrides

I. The pure Nb and Ta nitrides are prepared by synthesis from the elements. The reactant metals should be fine powders and should be degassed by heating in high vacuum. The N<sub>2</sub> must be completely free of O<sub>2</sub> and H<sub>2</sub>O. With fine metal powders, the reaction temperature should be 1200°C; with filings or solid metal, it must be 1300 to 1500°C. Temperatures higher than these naturally increase the nitridation rate; however, because of equipment limitations, they are employed only in the case in which the metal, in the form of wire, is clamped onto terminals and heated electrically in an  $N_2$  atmosphere.

Hydrides of Nb and Ta can be used instead of the metal. The hydrides lose their hydrogen during the first stages of the reaction, affording an especially reactive, fine metal powder; this, in turn, permits lower reaction temperatures. In addition, very pure NH<sub>3</sub> may be used instead of N<sub>2</sub>. The reaction of the metal with ammonia occurs at a temperature which is usually 300-400°C lower than that required for N<sub>2</sub>.

a) Nitridation of thin metal wire can be achieved at temperatures between 1350 and 2800 °C. This procedure, which uses nitrogen under pressure, always yields products with an N content corresponding to the upper limit, NbN. The rates of reaction are high but the amount of obtainable product is obviously small. Products with a low N content can be obtained by shortening the heating period and lowering the N<sub>2</sub> pressure; product homogeneity cannot be guaranteed, however.

b) To obtain nitrides from metal powder, the latter is placed in a sintered alumina boat inserted into a ceramic reactor tube. Because the nitrides are so extremely sensitive to oxygen, completely oxygen-free products can be obtained only if penetration of foreign gases is reduced by using the best, least porous ceramic reactor materials. Even these exhibit some porosity, however. The best method is to insert the reactor tube into another protective tube and fill the annular space between them with very pure N<sub>2</sub>. The products vary in N content depending on the temperature and duration of nitridation, and can range up to NbN or TaN. To achieve high homogeneity and nitrogen contents, the mixture of reactants must be cooled from time to time, removed from the apparatus, reground to a fine powder, and nitridized again.

The best method for obtaining products of a given desired N content is synthesis starting from a homogeneous mixture of highly nitridized materials and metal powder. Such mixtures are homogenized by prolonged calcination at at least 1400 °C (high vacuum or Ar atmosphere) with occasional cooling and regrinding of the calcined intermediate product.

II. Less pure niobium nitrides can be obtained from the oxide, carbon and nitrogen:

$$NbO_2 + 2C + \frac{1}{2}N_2 = NbN + 2CO$$
.

An intimate mixture of NbO<sub>2</sub> and the stoichiometric quantity of ash-free carbon is calcined in a stream of very pure N<sub>2</sub> or NH<sub>3</sub> at 1250°C. The products probably still contain some O and C.

This method is completely unsuitable for the preparation of Ta nitrides because in this case the product contains considerable quantities of O and C.

The nitrides can also be obtained from the oxides and  $NH_3$ , provided the reaction time is sufficiently long.

PROPERTIES:

Dark products with submetallic appearance. Products with a high N content are yellowish-gray or brown, those with a low N content are dark gray. M.p.: NbN about 2000 °C, TaN about 2800 °C. The N<sub>2</sub> decomposition pressures become appreciable at temperatures exceeding 1400 °C. Not attacked by acids. Readily and quantitatively converted to the pentoxides by moderate calcination in the presence of air (this is an analytical method).

The independent phases which exist in the Nb-N system correspond to the compositions  $NbN_{1.00-0.87}$ ,  $NbN_{0.79-0.75}$  and  $NbN_{0.50-0.40}$ ; density of these ranges from 8.3 to 8.4.

The independent phases in the Ta-N system correspond to the compositions  $TaN_{1,00}$  (d 13.8) and  $TaN_{0,50-0,40}$  (d 15.4).

REFERENCE:

- I a. K. Becker and F. Ebert. Z. Physik <u>31</u>, 269 (1925); K. Moers. Z. anorg. allg. Chem. <u>198</u>, 243 (1931); H. Rögener. Z. Physik <u>132</u>, 446 (1952).
- I b. C. Agte and K. Moers. Z. anorg. allg. Chem. <u>198</u>, 233 (1931);
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- II. E. Friederich and L. Sittig. Z. anorg. allg. Chem. <u>143</u>, 308 (1925).

## Niobium and Tantalum Phosphides

## NbP<sub>2</sub>, TaP<sub>2</sub>, NbP, TaP

These phosphides are prepared by synthesis from the elements, either in a "Faraday apparatus" (see Part I, p. 76) or by heating a mixture of metal powder and P in an  $Al_2O_3$  crucible. In either case the reactor is a sealed, evacuated quartz tube.

 $\begin{array}{l} Nb + P (2 P) = NbP (NbP_2) \\ 92.9 & 31.0 (62.0) & 123.9 (154.9) \\ Ta + P (2 P) = TaP (TaP_2) \\ 180.9 & 31.0 (62.0) & 211.9 (242.9) \end{array}$ 

In the "Faraday method," that end of the tube which contains the P is heated to 450-530 °C, while that containing the metal is heated first to 750 °C and then to 950 to 1100 °C.

In the method which uses a mixture, a fast onset of the reaction sometimes leads to explosion of the sealed tube, particularly if the P content of the reactant mixture is high.

The lower phosphides can also be obtained by degradation of products higher in phosphorus. This is done in high vacuum at 650-800°C.

PROPERTIES:

Black to dark-gray substances; fairly resistant to common reagents, vigorous reaction only with conc.  $H_2SO_4$ ; in the case of products high in phosphorus, also with conc. HNO<sub>3</sub>. Completely decomposed by fusion with alkaline oxidizing agents. d:  $TaP_2$  8.4; TaP 10.85.

**REFERENCE:** 

M. Zumbusch and W. Biltz. Z. anorg. allg. Chem. <u>246</u>, 35 (1941); A. Reinecke, F. Wiechmann, M. Zumbusch and W. Biltz. Z. anorg. allg. Chem. <u>249</u>, 1 (1942).

## Niobium and Tantalum Carbides

I. a) Pure products must be synthesized from the elements.

Nb + C = NbC; Ta + C = TaC 92.9 12.0 104.9 180.9 12.0 192.9

Intimate mixtures of Nb or Ta metal powders (or Nb or Ta hydrides with carbon (as ash-free as possible) are placed in graphite boats or graphite crucibles and heated in a vacuum or in an  $H_2$  atmosphere. Reaction temperatures vary between 1400 and 2100°C. When a tubular carbon furnace is used as the heat source and an  $H_2$  stream as the protective gas, the carbon content of the mixtures should be 15-20% lower than stoichiometric; this is

because the heating causes the  $H_2$  to react with the carbon of the furnace and the boat, and the resultant hydrocarbons supply the additional carbon needed to achieve the desired composition.

An additional purification of the carbide powder can be obtained by sintering (see TiN, p. 1233 f.). For purposes of presintering, the carbide powder is pressed into pellets which are embedded in loose carbide powder to protect them from chemical agents. These pellets are then presintered at a temperature of 2500-3000°C for about 15 minutes. The subsequent high-temperature sintering in argon at above 3000°C produces "self-purification" because of volatilization of impurities.

b) For small quantities of carbides, Nb (or Ta) wire is heated at temperatures exceeding 2500°C in an H<sub>2</sub> atmosphere which contains small amounts of hydrocarbon vapors. The presence of N<sub>2</sub> makes so little difference that up to 80% of the H<sub>2</sub> may be replaced by N<sub>2</sub>. Suitable hydrocarbons are toluene, methane and acetylene. The nascent carbides formed may lose carbon at the high temperatures if the CH<sub>4</sub> content of the gas is less than about  $^{1}/_{4}$ %, or that of C<sub>2</sub>H<sub>2</sub> is less than about  $^{1}/_{8}$ %.

II. The carbides can also be obtained by reacting the oxide with carbon.

 $\begin{array}{rrrr} NbO_2 &+ & 3 \ C &= & NbC &+ & 2 \ CO \\ 124.9 & & 36.0 & & 104.9 & & 56.0 \\ Ta_2O_5 &+ & 7 \ C &= & 2 \ TaC &+ & 5 \ CO \\ 441.9 & & 84.1 & & 385.9 & & 140.1 \end{array}$ 

The respective powder mixtures, in Mo or carbon boats, are reacted at 1250-2300 °C in an H<sub>2</sub> stream. At 2300 °C, a one-hour heating is recommended. When carbon tubes or boats are used as in method Ia, the mixture may be less than stoichiometric with respect to carbon. The products may be further purified via the above-cited sintering.

Alternate methods: Crystal growing procedures.

a) A carbon fiber is resistance-heated to above 2000°C with current. The fiber is in a "reaction lamp," which also holds thoroughly degassed  $TaCl_5$  (this process cannot be used with Nb). The quantity of  $TaCl_5$  needed is difficult to measure out (also, with excess chloride, free metal or lower carbides are formed on the incandescent fiber). At any rate, this process may be followed by carbidization in the presence of  $H_2$  and a hydrocarbon vapor [W. G. Burgers and J. C. M. Basart, Z. anorg. allg. Chem. <u>216</u>, 207 (1934)].

b) Carbide may also be deposited on a tungsten wire exposed at 1900-2300 °C and 0.1 mm. Hg to an H<sub>2</sub> carrier gas containing small amounts of  $TaCl_5$  (NbCl<sub>5</sub>) and toluene vapor. However, this

carbide product will contain a large amount of free metal and must be subjected to a postcarbidization treatment [K. Becker and H. Ewest, Z. techn. Physik <u>11</u>, 148 (1930); K. Moers, Z. anorg. allg. Chem. <u>198</u>, 243 (1931)].

c) According to a patent [D. Gardner, U.S. Patent 2,532,295 (1946/50)], the pentachlorides can be reacted with  $H_2$  and carbon derivatives such as CCl<sub>4</sub> or CaC<sub>2</sub> even at 600 to 700 °C.

#### LOWER CARBIDES

Most preparative methods describe the synthesis of carbides of the limiting composition NbC and TaC. However, method I (or, if properly executed, also the above-described crystal growing procedure) also gives products with a low C content, e.g., products corresponding to the lower carbides Nb<sub>2</sub>C (and Ta<sub>2</sub>C).

PROPERTIES:

Iron-gray to dark-gray powders; the sintered solid exhibits a bright metallic luster; tarnishing frequently changes the surface color to brown to yellow. Does not lose carbon at high temperatures in the presence of hydrogen, provided a small amount of hydrocarbons is present in the gas (see method Ib). Stable to  $N_2$  up to about 3300°C. Quite sensitive to  $O_2$  and  $H_2O$  on heating, undergoing rapid oxidation above 800°C in air. Not very volatile in high vacuum up to 3000°C.

NbC: m.p. 3500°C; d 7.6. TaC: m.p. 3900°C; d 13.9.

REFERENCE:

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  G. Brauer and R. Lesser. Z. Metallkunde <u>49</u>, 622 (1958); <u>50</u>, 8 (1959);
  E. K. Storms and N. H. Krikorian. J. Phys. Chem. <u>63</u>, 1747 (1959);
  R. P. Elliott. A.S.M. Preprint No. 179 (1960).
- Ia, b. K. Becker and H. Dwes. Z. techn. Physik 11, 148 (1930).
- II. E. Friederich and L. Sittig. Z. anorg. allg. Chem. <u>144</u>, 169 (1925); C. Agte and K. Moers. Z. anorg. allg. Chem. <u>198</u>, 233 (1931); A. Y. Kovalskiy and Y. S. Umanskiy. Zh. Fizich. Khimii <u>20</u>, 769 (1946); Chem. Zentr. <u>47</u>, II, 1546.

## **SECTION 24**

Chromium, Molybdenum, Tungsten, Uranium

## F. HEIN and S. HERZOG

#### Chromium

Cr

I.

 $Cr_2O_3 + 2 Al = 2 Cr + Al_2O_3$ 152.0 53.9 104.0 101.9

An intimate mixture of 70 g. of pure ignited  $Cr_2O_3$ , 33 g. of Al granules (or Al powder), and 25 g. of fused and powdered  $K_2Cr_2O_7$  is placed in a clay crucible whose bottom is covered with 10 g. of  $CaF_2$ . The mixture is caused to react by means of ignition mixture and a strip of Mg.\* After cooling, the contents of the crucible are broken up, and the spheres of metal are mechanically extracted. This gives about 99% pure chromium metal in 50-75% yield (the larger the quantity of reactants, the better the yield).

## II. $2 K_2[CrCl_5(H_2O)] + 3 Mg = 2 Cr + 3 MgCl_2 + 4 KCl + 2 H_2O$ 651.0 37.0 104.0

The  $K_2[CrCl_5(H_2O)]$  is obtained by dissolving 100 g. of  $K_2Cr_2O_7$ in the minimum quantity of water, treating the solution with 400 ml. of HCl (d 1.124), and gradually adding 100 ml. of 80% alcohol. The reaction is accompanied by vigorous evolution of heat. Then, 170 g. of KCl is added and completely dissolved, the mixture filtered, and the filtrate evaporated to dryness. The mass is then completely dehydrated by further heating. The resulting violet solid is then

<sup>\*</sup>The ignition mixture, called "Zündgemisch" or "Zündkirsche" (igniting cherry), consists of an intimate mixture of 15 parts by weight of barium peroxide and 2 parts of powdered magnesium metal held together with collodion. The whole is wrapped in magnesium ribbon, which acts as fuse (H. Blücher, Auskunftsbuch für die chemische Industrie [Data Book for the Chemical Industry], 18th ed., de Gruyter, Berlin, 1954, p. 1314).

ground, and any green portions are removed as completely as possible. The potassium chromium (III) chloride thus obtained is mixed with 50 g. of Mg filings. The mixture is placed in a covered Hessian crucible, brought to red heat and held at this temperature for one half hour. However, not all of the KCl must be allowed to volatilize, otherwise a fraction of the Cr will be converted to the oxide and will contaminate the product. After this calcination, the crucible is cooled and broken. The shards and the particles of green chromium oxide, which appear on the surface of the grayblack melt, are removed. The mechanically cleaned mass is then placed in water, where it crumbles to a powder. The soluble salts are removed by decantation. The residue is boiled with dilute nitric acid to remove the excess Mg (and the MgO which has formed from it).

Any  $Mg(NO_3)_2$  and excess acid present are separated by further decantation; filtration is not recommended because of the fine particle size of the metal. The Crresidue is dried on a steam bath. Yield about 27 g. of light-gray, microcrystalline powder whose Cr content is 99.6%.

#### III. ELECTROLYSIS

The electrolytic cell consists of a beaker with a copper cathode rod suspended in the center. A lead sheet or a coil of lead tubing placed along the wall of the cell serves as the anode; if the latter arrangement is used, cold water is circulated through the tubing. The electrolyte consists of a solution of 240 g. of  $CrO_3$ , 3 g. of  $Cr_2(SO_4)_3 \cdot 12 H_2O$ , and 8.8 g. of  $Cr(OH)_3 \cdot 3 H_2O$  in one liter of water. A current density of 0.10 amp./cm.<sup>2</sup> and a potential of 3.2 volts are used. It is essential that the electrolyte remain undisturbed (no stirring) during electrolysis. The thick layer of Cr which forms on the cathode in a few days is readily stripped off. If it should prove necessary to remove the  $H_2$  which accumulates in the metal voids during deposition of the Cr, the product should be heated to 600°C in high vacuum.

IV. Ductile Cr is obtained from  $CrCl_3$  and Ca in a steel bomb under argon. For details of the method, see section on Titanium, p. 1161.

#### PROPERTIES:

Atomic weight 52.01. Solid Cr has a silvery luster; very hard and brittle, but very pure Cr is ductile. M.p. about 1920°C (in vacuum), b.p. about 2200°C;  $d^{25}$  7.138. Body-centered cubic crystals; hexagonal form also exists.

**REFERENCES:** 

- I. G. Jander, Lehrbuch für das anorg. chem. Praktikum [Textbook of Inorg. Chem. Practice], 5th ed., 1944, p. 188.
- II. H. Funk. Darst. der Metalle im Laboratorium [Preparation of Metals in the Laboratory], Stuttgart, 1938, p. 66.
- III. Private communication from Prof. G. Grube. See H. Haraldsen and E. Kowalsky. Z. anorg. allg. Chem. <u>224</u>, 330 (1935).
- IV. W. Kroll. Z. anorg. allg. Chem. 226, 23 (1935).

## Chromium (II) Chloride

#### CrCl<sub>2</sub>

I.

 $CrCl_3 + \frac{1}{2}H_2 = CrCl_2 + HCl_158.4$  11.2 l. 122.9

The special vessel  $(2.5 \times 50 \text{ cm.}, \text{ Fig. 313})$  used for the reduction is made of high-melting glass. For reasons of safety it is first baked in vacuum at 500°C while empty. It is provided with a twohole rubber stopper, through which the outlet tube *a* and the longer inlet tube *b* (8-mm. diameter) are inserted. The reactor is charged with CrCl<sub>3</sub> (prepurified by sublimation in a stream of Cl<sub>2</sub>). The tube is heated to 500°C in a thermostatically controlled electric



Fig. 313. Preparation of chromium (II) chloride. b inlet tube for hydrogen, H<sub>2</sub>-HCl mixture, or nitrogen; d stopcock with10-mm. bore for withdrawal of reaction product.

furnace, while a mixture of  $H_2$  and HCl (the latter serving to hinder any further reduction to Cr) is admitted through b (the dry,  $O_2$ -free, 50 ml./min. gas streams are mixed in a tee prior to introduction into the apparatus). The outlet tube a is connected to a drying tube filled with CaCl<sub>2</sub>. To test for completeness of reduction, the furnace is removed briefly from time to time (pure CrCl<sub>2</sub> is white). When the reaction is complete, the furnace is cooled, the H<sub>2</sub>-HCl mixture displaced with dry N<sub>2</sub> or CO<sub>2</sub>, and the inlet tube pulled back until its tip is at the rubber stopper. The reactor tube is then melt-sealed at constriction c. Any required quantity of CrCl<sub>2</sub> can be shaken out of the tube through the 10-mm.-bore stopcock d; in this operation, the CrCl<sub>2</sub> must always be under dry, O<sub>2</sub>-free inert gas, which is admitted through stopcock e. (For special apparatus for storage under inert gas, see also Part I, pp. 71 and 75.)

II.

$$Cr + 2 HCl = CrCl_2 + H_2$$
  
52.0 44.5 l. 122.9

A small porcelain boat is charged with pea-sized pieces (or, better, powder) of metallic Cr and inserted in a quartz reactor tube. Dry,  $O_2$ -free HCl is passed through the tube, which is heated to as high a temperature as possible (1150 to 1200°C). On cooling in the HCl stream, an asbestoslike mass of white (or, if impure, gray) crystalline needles of  $CrCl_2$  is obtained. Because of its toughness, the mass is very difficult to remove from the boat. The preparation must be sealed as rapidly as possible into a sample tube filled with  $N_2$  or  $CO_2$ ; if this is not done, the anhydrous  $CrCl_2$  is rapidly hydrated by atmospheric moisture, after which oxidation also occurs at once. Because of its high melting point, some metal may be trapped within the chloride and thus remain unreacted.

 $\begin{array}{c} \text{ether} \\ \text{III.} \\ \text{Cr}_2(\text{CH}_3\text{COO})_4 + 4 \text{HCl} = 2 \text{CrCl}_2 + 4 \text{CH}_3\text{COOH} \\ 340.2 \\ 145.8 \\ 245.9 \end{array}$ 

Ten grams of fine chromium (II) acetate hydrate  $Cr_2(CH_3COO)_4$  · 2 H<sub>2</sub>O crystals is dehydrated in a three-neck flash at 110 to 120 °C (aspirator vacuum); the color changes from brick-red to brown. Then, 60 ml. of air-free ether is added and dry HCl is passed through the vessel with the suspension (the vessel is in an ice bath and is protected against atmospheric moisture by a P<sub>2</sub>O<sub>5</sub> tube). After several minutes, a violet color is observed in the solution, and the chromium acetate powder is transformed (with an increase in bulk) into chromium (II) chloride which still contains some acetic acid. The flask is swirled during this operation to prevent clogging of the inlet tube. The white crystals are filtered in the absence of air, washed with absolute, air-free ether, and dried at 110 to 120 °C. The acetic acid is thus eliminated, and pure white, analytically pure  $CrCl_2$  is obtained as a residue. Yield 4-5 g.

PROPERTIES:

White crystals or fused, fibrous mass. M.p. 824 °C;  $d^{14}_{4}$  2.751. Very hygroscopic. Can be sublimed in vacuum. Dissolves readily in water, giving a sky-blue solution.

1337

**REFERENCES:** 

- I. J. Reschke. Thesis, Univ. of Leipzig, 1925, p. 43; F. Hein. Z. anorg. allg. Chem. <u>201</u>, 314 (1931); F. Ephraim. Helv. Chim. Acta <u>17</u>, 291 (1934); A. B. Burg in: L. F. Audrieth. Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 150.
- II. J. Koppel. Z. anorg. Chem. <u>45</u>, 361 (1905); W. Biltz and E. Birk. Z. anorg. allg. Chem. <u>134</u>, 134 (1924); W. Fischer. Z. anorg. allg. Chem. <u>222</u>, 309 (1935); H. Hecht. Präparative Anorganische Chemie [Preparative Inorganic Chemistry], Berlin-Göttingen-Heidelberg, 1951, p. 80.
- III. Private communication from F. Hein, E. Kurras and W. Kleinwächter (unpublished).

## Chromium (III) Chloride

## CrCl<sub>3</sub>

I.

 $2 Cr + 3 Cl_2 = 2 CrCl_3$ 104.0 66.0 l. 316.8

Coarse Cr metal powder (10-20 g.) is placed in a 50-cm.-long and 3-cm.-I.D. horizontal porcelain reactor tube, which is heated in a blast lamp flame. It is essential that all residual air be displaced by a fast stream of completely dry  $Cl_2$  (for at least half an hour prior to introduction of the Cr). The temperature is then raised as high as possible; the tube is allowed to cool, and the  $Cl_2$  is displaced with dry  $CO_2$ . Violet leaflets of CrCl<sub>3</sub> form, with a large increase in volume (to avoid plugging the tube as a result of this volume increase, the Cr reactant should be distributed over a long stretch of the tube).

The  $CrCl_3$  is purified by sublimation in the stream of  $Cl_2$ , then repeatedly boiled with conc. HCl, washed with distilled water until disappearance of chloride reaction, and dried at 200 to 250 °C.

The high sublimation temperature of over  $1200^{\circ}$ C is a disadvantage of this method. Under certain conditions the procelain tube can be markedly corroded in this operation and, in addition, it may be plugged as a result of the large volume increase occurring during the formation of the chromium (III) chloride. The following method, in which green chromium (III) chloride hydrate is dehydrated in a stream of CCl<sub>4</sub> at 600°C, avoids these difficulties.

**II.** OVERALL EQUATION :

 $[CrCl_2(OH_2)_4]Cl \cdot 2H_2O = CrCl_3 + 6H_2O$ 

SIDE AND INTERMEDIATE REACTIONS :

$$2[CrCl_{2}(OH_{2})_{4}]Cl \cdot 2 H_{2}O = Cr_{2}O_{3} + 6 HCl + 9 H_{2}O$$
  

$$Cr_{2}O_{3} + 3 CCl_{4} = 2 CrCl_{3} + 3 COCl_{2}$$
  

$$2 Cr_{2}O_{3} + CCl_{4} = 4 CrCl_{3} + CO_{2}$$

Thus, one by-product is phosgene, which must be carefully vented (use a hood!). Simple absorption in water is not adequate, since the hydrolysis is not instantaneous.

The apparatus is shown in Fig. 314. Forty grams of green chromic chloride is placed in quartz flask c, which is placed in an electric furnace capable of delivering 650°C. The apparatus for generating and superheating CCl<sub>4</sub> vapor is then attached. This apparatus consists of the 250-ml. distilling flask a with its superposed dropping funnel and a U tube immersed in a silicone oil bath at 150°C.



Fig. 314. Dehydration of chromium (III) chloride hydrate with carbon tetrachloride.

After brief heating of the furnace (flask temperature of 100-150°C), one drop of  $CCl_4$  per second is admitted from the dropping funnel into flask *a*, which is heated with a Bunsen burner in such a manner that each drop vaporizes at once. After some time (furnace temperature of about 300°C), a mixture of water,  $CCl_4$ , etc., distills over. It condenses in *d*; the noncondensing gases (including phosgene) are vented through the hood. After some two hours of reaction, when the furnace temperature has reached 650°C, the gas stream is interrupted and the apparatus allowed to cool. The anhydrous  $CrCl_3$  remains in the flask in about 90% yield (some of it sublimes). The lustrous violet crystalline leaflets are extracted with boiling dilute HCl and dried. About 20 g. of chromium (III) chloride is obtained. PROPERTIES:

Formula weight 158.38. Red-violet crystalline scales with metallic luster. M.p. ~ 1150 °C. May be sublimed in  $Cl_2$  stream; rate of solution in water, acids and organic solvents immeasurably slow. Addition of a very small amount of  $CrCl_2$  aids in rapid solution of  $CrCl_3$  in water or alcohol.

**REFERENCES:** 

- I. Private communication from H. Hecht, Greifswald; F. Hein. Z. anorg. allg. Chem. 201, 314 (1931).
- II. G. B. Heisig, B. Fawkes and R. Hedin in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 193; A. Vavoulis et al. Ibid., Vol. VI, New York-London, 1960, p. 129.

#### Chromium (II) Bromide

#### CrBr<sub>2</sub>

I.

 $CrBr_3 + 0.5H_2 = CrBr_2 + HBr$ 291.8 11.2 l. 211.8

A weighed quantity of  $CrBr_3$  is reduced to constant weight in a U tube at 350-400 °C for 6-10 hours. The most painstaking purification of the  $H_2$  is essential for success. For uniform heating, the U tube is surrounded with an asbestos box.

II.  $Cr_{2}(CH_{3}COO)_{4} + 4 HBr = 2 CrBr_{2} + 4 CH_{3}COOH$ 340.2 323.7 423.7

The procedure is the same as for chromium (II) chloride (method III), but hydrogen bromide is used instead of hydrogen chloride.

PROPERTIES:

White crystalline powder;  $d_2^{25}$  4.356. Soluble in air-free water, yielding a blue color. Rapidly oxidized in air.

**REFERENCES**:

- I. W. Biltz and E. Birk. Z. anorg. allg. Chem. <u>134</u>, 134 (1924);
   W. Fischer and R. Gewehr. Ibid. <u>222</u>, 309 (1935).
- II. Private communication from F. Hein, E. Kurras and W. Kleinwächter (unpublished).

#### Chromium (III) Bromide

#### CrBr<sub>3</sub>

Electrolytic Cr powder is spread in a thin layer in a quartz reactor tube and bromine vapor, predried over  $P_2O_5$ , is passed over the Cr in a stream of  $O_2$ -free, dry  $N_2$  or Ar. The quartz tube is heated to about 1000 °C, whereupon the two elements combine with incandescence at the beginning of the reaction. The unconverted  $Br_2$  is condensed in a receiver cooled in ice-salt mixture. After 45-60 minutes, the material is allowed to cool while maintaining the gas stream. The lustrous black, leafy crystals of CrBr<sub>3</sub> are purified by several extractions with absolute ether and decantations with ice water (to remove traces of adhering  $CrBr_2$ ). This is followed by washing with absolute alcohol and ether; the product is dried in a vacuum desiccator over  $P_2O_5$ .

PROPERTIES:

Formula weight 291.76. Black lustrous crystals, green in transmitted and reddish in reflected light. Soluble in water only upon addition of Cr (II) salts.

REFERENCES:

J. Reschke. Thesis, Univ. of Leipzig, 1925; F. Hein and I. Wintner-Hölder. Z. anorg. allg. Chem. <u>201</u>, 319 (1931).

## Chromium (II) Iodide

#### $CrI_2$

 $Cr + I_2 = CrI_2$ 52.0 253.8 305.9

The starting materials are electrolytic Cr reduced to the size of millet seed and  $I_2$  resublimed over KI (see p. 277). The apparatus is shown in Fig. 315. The quartz reactor tube *a* has an overall length of about 56 cm. and an I.D. of 2 cm. (2.5 cm. at the bulged-out section). The tube ends (on the right side of the drawing) in a quartz capillary spiral which is somewhat constricted at q; on the opposite side, it can be closed off by a large-diameter ground joint connected to stopcock  $h_1$ . The cooling trap k is connected via stopcock  $h_2$  to a tee adapter; thus, it can be in line either with the high-vacuum system (via stopcock  $h_3$ ) or with the two-way stopcock z, which in turn leads either to the N<sub>2</sub> source or to a vacuum oil pump.

The movable tubular resistance furnace *o* serves to heat the quartz reactor at any desired spot.



Fig. 315. Preparation of chromium (II) iodide. a quartz reactor tube; s boat with Cr; o electric furnace; k cold trap.

First, a trace of very fine Cr powder is placed in the tip of the capillary spiral, and then  $I_2$  (10% excess over the stoichiometric quantity) is placed in the slightly inclined tube section *i*. A small porcelain boat with the Cr is placed in the bulge *s*. Trap *k* is now cooled to -80°C and the apparatus is evacuated (with stopcock *z* closed) to < 0.001 mm.

The vacuum is then broken with  $N_2$  and the evacuation repeated (this procedure is repeated 3 or 4 times). Stopcock  $h_1$  is closed and the Cr powder at the tip of the spiral is heated with a torch to a bright red glow to bind the last traces of O<sub>2</sub>. Then the tube furnace is set in place over spot s and the Cr heated at 700 to 850°C: to minimize undesirable heat losses, asbestos paper (not shown in the figure) is placed in both furnace openings. As the I<sub>2</sub> now diffuses slowly toward the Cr, the nascent CrI<sub>2</sub> solidifies as a crystalline mass at both sides of the tube protruding from the furnace. This slow procedure yields very beautiful leaf- or needle-shaped redbrown to iodine-colored crystals. At the beginning of the iodination, a deep-black coating always forms on the colder portions of the tube; this material is more volatile than CrI<sub>2</sub> and converts to CrI<sub>2</sub> at higher temperatures, evolving iodine. Even though the CrI<sub>2</sub> itself is not particularly volatile, the crystal deposit extends on both sides for up to 2 cm. beyond the hot zone. By gradually shifting the furnace, the crystalline deposit is shifted away from the spot where the Cr is situated; in this way, one prevents it from becoming too dense and provides a surface for fresh deposition. By following this procedure, almost all the I<sub>2</sub> is introduced little by little, with the long heating producing a type of resublimation and yielding beautiful single crystals. Such a run takes 8-14 days; some unreacted metal is still invariably found after this time at the bottom of the boat, because the sublimation of  $CrI_2$  from the (protective) melt is very slow. When the iodination has proceeded as far as possible, the excess  $I_2$  must be displaced before the  $CrI_2$  itself can be removed. Thus, the reactor is connected to the high vacuum (via  $h_3$ ), with the cold trap at -80°C. The  $I_2$ is driven off by careful fanning with a flame (250-300°C) and continuous evacuation. Evacuation is continued after complete removal of the  $I_2$  until the quartz tube is cool; the apparatus is then closed off at  $h_1$ .

Transfer of the  $CrI_2$  from the reactor requires great care, since the compound is extremely sensitive to air and moisture. Further, the solid material possesses a relatively large surface. The following procedure is the safest: The vacuum is broken with compressed  $N_2$  via  $h_1$ , and pressure above atmospheric is created in the apparatus. The quartz capillary is then broken off at q and this opening is attached to another  $N_2$  connection. The



Fig. 316. Transfer of chromium (II) iodide following its preparation. *a* reaction tube of Fig. 315; *p* porcelain boat with front wall broken off (scoop).

joint with  $h_1$  is then removed and the adapter tube b is inserted in its place, as shown in Fig. 316 (an N<sub>2</sub> stream also passes through b). The adapter carries at one end the same standard taper joint as  $h_1$ , and at the other end two short connectors m and n. An N<sub>2</sub> stream enters at n; a long glass rod, the forward end of which is bent into a small hook, is introduced through m. This hook supports a fairly large porcelain boat p, whose front wall is broken off at a sharp angle. The boat is of such a size that it is still able to move in the quartz reactor. Now, with the fast  $N_2$  stream always maintained, this "dredge" is pushed forward through the loose crystal aggregate, whereupon most of the latter drops into boat s. This operation is aided by gentle tilting and tapping of the reactor. The filled boat s is now pulled back into the adapter b; the adapter is quickly detached from the reactor and closed off with a ground cap. Finally the boat with the CrI<sub>2</sub> is transferred for storage into a tube provided with a ground joint (see PartI, p. 75); as a precaution, this tube is evacuated several times and then filled with  $N_2$  (slight gage pressure). All of these operations must be carried out in sequence and without undue delay.

PROPERTIES:

Brown-red leaflets; thin leaflets are somewhat transparent, thick crusts often have an iodinelike color. Very sensitive to air and moisture, easily soluble, with bright blue color, in air-free water. M.p. 790-795 °C;  $d_{5}^{20}$  5.02<sub>3</sub>.

REFERENCE:

F. Hein and G. Bähr. Z. anorg. allg. Chem. 251, 241 (1943).

#### Chromium (III) Iodide

CrI<sub>3</sub>

The apparatus (Fig. 317) is made of high-melting glass. The diameter of the tube at e, f and h is 25-30 mm. Three grams of fine chromium metal powder (electrolytic chromium is best) is introduced into e, and an excess (30 g.) of iodine into h. Then k is sealed off and the apparatus is evacuated to appr.  $10^{-5}$  mm. (with occasional gentle heating). The iodine is now sublimed into f by cooling the latter section (Dry Ice) and heating h (during this operation stopcock b is turned off and on). Following this, g is sealed off in high vacuum, and then d is sealed off as well.



Fig. 317. Apparatus for synthesis of chromium (III) iodide.

Now sections e and f are enclosed in separate tubular furnaces; e is heated for 24 hours at 475°C, and f for the same period at 225°C. (The vapor pressure of iodine at this temperature is approximately 3 atm.) The apparatus is then allowed to cool, and the unreacted iodine is sublimed out from e (which is held at 100°C) into f (held at room temperature). The tube is then broken, in dry

1344

air, at constriction l (careful!). The material is transferred to a suitable storage vessel, which is then evacuated for some time to remove the last traces of iodine. The yield is almost quantitative.

If necessary, further purification may be achieved by heating for several hours at 500 °C in an evacuated quartz tube. This operation yields chromium (II) iodide; the iodide can be sublimed in vacuum at 700 °C and finally reiodinated as described above.

PROPERTIES:

Black crystals; dissolve rapidly in water upon addition of some Cr (II) iodide. Stable at room temperature; thermally dissociated at higher temperatures according to:  $2 \text{ CrI}_3 = 2 \text{ CrI}_2 + I_2$ . The iodine pressure reaches 1 atm. at about 670 °C.

REFERENCES:

L. L. Handy and N. W. Gregory. J. Amer. Chem. Soc. <u>72</u>, 5049 (1950); N. W. Gregory and L. L. Handy in: T. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 128.

## Chromium (III) Hydroxide

#### Cr(OH)<sub>3</sub> · n H<sub>2</sub>O

I. A-HYDROXIDE  $Cr(OH)_3 \cdot 3 H_2O$ 

A solution of 12 g. of gray-blue  $[Cr(H_2O)_6]Cl_3$  (for preparation, see p. 1348 f.) in 500 ml. of water is treated with 100 ml. of 2 N ammonia. After settling of the precipitate (centrifuge if necessary), the mother liquor is decanted; the suspension is filtered through a leaf filter, thoroughly washed until free of NH<sub>4</sub>Cl, and dried in air.

PROPERTIES:

Bright blue-green powder; gives blue salts of the  $[Cr(H_2O)_6]X_3$  type with dilute acids.

B-HYDROXIDE  $Cr(OH)_3 \cdot 3 H_2O$ 

The procedure for the preparation of the A-hydroxide is followed, but one starts with 12 g. green of  $[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O$ .

PROPERTIES:

Dark blue-green powder; gives green salts of the  $[Cr(H_2O)_4X_2]X \cdot 2 H_2O$  type with dilute acids. In contrast to the A-type, the B-hydroxide is insoluble in acetic acid.

#### F. HEIN AND S. HERZOG

# 1346

$$\begin{array}{rcl} \text{II.} & 2 \operatorname{CrO}_3 + 3 \operatorname{C}_2 \text{H}_5 \text{OH} = 2 \operatorname{Cr}(\text{OH})_3 + 3 \operatorname{CH}_3 \text{CHO} + 3 \operatorname{H}_2 \text{O} \\ & (\cdot n \operatorname{H}_2 \text{O}) \end{array}$$

This is a convenient method for preparation of larger quantities of  $Cr(OH)_3$ . A solution of 160 g. of  $CrO_3$  in 2 liters of water is prepared and alcohol (8 portions of 10 ml. at 5-minute intervals) is added with vigorous stirring (caution: a hood is needed!). After 4 hours of standing, an additional 80 ml. of alcohol is added in the same manner. The mixture is then refluxed for 16 hours (stirring is required to avoid bumping). The finely divided, dark brown precipitate is filtered through a 24-cm.-diameter Büchner funnel and dried at 110°C without washing. Yield: 145 to 150 g. Additional quantities (30-35 g.) can be recovered from the filtrate by concentration of the latter. Alternately, the Cr-containing liquid may be used as solvent (instead of water) in the next run.

#### PROPERTIES:

This method affords a black product with a pitchlike luster, probably because of a small admixture of higher oxides. This material has a higher catalytic activity than that obtained by precipitation.

**REFERENCES:** 

I. A. Hantzsch and E. Torke. Z. anorg. allg. Chem. <u>209</u>, 73 (1932).
II. R. F. Ruthruff in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 190.

## Chromium Sulfides

CrS, Cr<sub>2</sub>S<sub>3</sub>

I.

Cr + S = CrS52.0 32.1 84.1 2 Cr + 3 S = Cr<sub>2</sub>S<sub>3</sub> 104.0 96.2 200.2

The sulfides are prepared by heating exact stoichiometric mixtures of electrolytic Cr (for preparation, see p. 1335) and pure S for 24 hours in small, evacuated, sealed quartz tubes placed in an electric furnace at 1000°C. All of the S does not react even if heated for 3-4 days and slowly cooled. The product is freed of unreacted sulfur by fanning with a Bunsen flame while simultaneously cooling the empty seal-off point of the tube. The quantity of S which condenses in that section is determined by reweighing, and the composition of the sulfide is calculated by using this value.

Cracking of the quartz tube during cooling of preparations which are high in S can be avoided by sealing the reaction vessel proper in a second, similarly evacuated quartz tube.

II.  $2 \operatorname{CrCl}_3 + 3 \operatorname{H}_2 S = \operatorname{Cr}_2 S_3 + 6 \operatorname{HCl}_{316.8} 66.4 l. 200.2$ 

Exactly stoichiometric  $Cr_2S_3$  may be obtained by heating  $CrCl_3$  in a stream of  $H_2S$  at 600-650°C.

PROPERTIES:

The chromium sulfide preparations obtained via method I have a metallic appearance and become fused at the temperature of preparation. The CrS possesses a hexagonal superstructure of the B 8 type, while at 59.7 atom% of S, a B 8 structure with the axial ratio  $c/a = 1.62_5$  has been shown to exist. The Cr<sub>2</sub>S<sub>3</sub> obtained by method II consists of hexagonal black leaflets, resistant to nonoxidizing acids, easily soluble in HNO<sub>3</sub>.

**REFERENCES**:

- I. H. Haraldsen and E. Kowalsky. Z. anorg. allg. Chem. <u>224</u>, 331 (1935); H. Haraldsen and A. Neuber. Ibid. <u>234</u>, 338 (1937).
- II. W. Rüdorff and E. Stegemann. Z. anorg. allg. Chem. <u>251</u>, 390 (1943).

## Chromium Nitride

#### CrN

I.

$$Cr + \frac{1}{2}N_2 = CrN$$
  
52.0 11.4 l. 66.0

Electrolytic chromium powder is heated for 2 hours at 800-900°C in a quartz or porcelain tube while a dry,  $O_2$ -free stream of  $N_2$  is passed through. After cooling, the product is ground in an agate mortar and calcined again for 2 hours in a stream of  $N_2$ . The final product is treated with HCl until nothing further dissolves (the HCl liquid remains colorless). The black residue is thoroughly washed and dried. 1348

II.

 $CrCl_3 + 4 NH_3 = CrN + 3 NH_4Cl$ 158.4 88.3 l. 66.0

A tube of high-melting glass (25-30 cm. long) is used and 5-10 g. of anhydrous CrCl<sub>3</sub> is calcined, first gently and then vigorously, in a stream of NH<sub>3</sub>. The heat source is a series of burners. The NH<sub>3</sub> is obtained either from a bomb or by heating about 300 ml. of conc. ammonia; it is dried by passage through a lime tower and a large U tube filled with CaO. The reactor tube carries no outlet tube, since the latter would be plugged by sublimed NH<sub>4</sub>Cl (use a hood!). Strong heating is continued until no further NH<sub>4</sub>Cl vapor is evolved; then (after cooling) the product is ground and recalcined in a stream of NH<sub>3</sub>. The yield is almost quantitative.

If it is desired to remove traces of  $CrCl_3$ , the product is extracted in the cold with some dilute HCl (add some Sn), then washed with water, filtered and dried at 100-120°C.

PROPERTIES:

Black, magnetic powder; insoluble in acids and alkalies; d 5.9. Crystal structure: NaCl type.

REFERENCES:

- I. F. Briegleb and A. Geuther. Liebigs Ann. <u>123</u>, 239 (1862); R. Blix. Z. phys. Chem. B 3, 236 (1929).
- II. H. Biltz and W. Biltz. Übungsbeispiele aus der unorg. Experimentalchemie [Exercises in Experimental Inorganic Chemistry], 3rd and 4th eds., 1920, p. 20.

## Hexaaquochromium (III) Chloride

#### $[Cr(OH_2)_6]Cl_3$

I.  $[Cr(OH_2)_6] (NO_3)_3 + 3 HCl = [Cr(OH_2)_6]Cl_3 + 3 HNO_3$  $(\cdot 3 H_2O)$ 400.2 266.5

A solution of 100 g. of chromium (III) nitrate  $Cr(NO_3)_3 \cdot 9 H_2O$ in 100 ml. of  $H_2O$  and 100 ml. of 38% HCl is prepared. Hydrogen chloride gas, predried in  $H_2SO_4$ , is introduced with ice cooling, until the precipitation of  $[Cr(H_2O)_6]Cl_3$  is complete.

The crystal slurry is rapidly filtered on a large glass suction funnel and washed with some fuming HCl. It is dissolved in 100 ml. of water and 100 ml. of fuming HCl, and reprecipitated with HCl gas while cooling in ice. After precipitation is complete, the greenish supernatant solution is decanted and the gray-blue chloride is freed of most of the adhering HCl and the green chloride by stirring three times with acetone. The remaining impurities are completely extracted by treating the product with small quantities of acetone on a fritted-glass funnel (the filtrate must become colorless in the end). The acetone is removed by rinsing with absolute ether. The salt is freed of ether and traces of moisture by drying in a desiccator over  $H_2SO_4$ . The yield is about 72%.

II. 
$$[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O = [Cr(H_2O)_6]Cl_3$$
  
266.5 266.5

A solution of 50 g. of green chromium chloride hydrate in 50 g. of water is refluxed for one half hour, during which time almost no color change is observed. The flask is then cooled by immersion in an ice-salt mixture and HCl gas is introduced with periodic shaking of the flask. The temperature inside the flask must always be held below 0°C; this is achieved by frequent renewal of the freezing mixture. After saturation with HCl, the fine powder which separates is allowed to settle to the bottom and the supernatant blue-green liquid is decanted. The powder itself is rinsed out of the flask onto a fritted glass funnel with cold saturated HCl, dried as much as possible by suction, then stirred with acetone and washed until the acetone is no longer green. As soon as the acetone traces have evaporated, the crude product is dissolved in 20 ml. of water; it is filtered if necessary, and HCl gas is introduced into the blue solution (while cooling the flask with cold water) until saturation. At this point the gas flow is interrupted and the flask is placed in finely crushed ice. The solution becomes almost colorless after some time while the chloride separates in granular, bluegray crystals. After filtering through a fritted-glass funnel, the product is washed with acetone and dried over H<sub>2</sub>SO<sub>4</sub>. Yield: 12 g.

III. 
$$\operatorname{KCr}(\operatorname{SO}_4)_2 \cdot 12 \operatorname{H}_2O + 3 \operatorname{HCl} \rightarrow [\operatorname{Cr}(\operatorname{H}_2O)_6]\operatorname{Cl}_3$$

Chrome alum (250 g.) is dissolved in a chilled mixture of 1 liter of conc. hydrochloric acid and 250 ml. of water. The solution is filtered and saturated with hydrogen chloride gas at 10 to  $15^{\circ}$ C. It becomes almost colorless during this step, and the crude product separates in crystalline form. The crystals are filtered and purified by dissolving in 175 ml. of water, reprecipitating at  $10^{\circ}$ C as described above, filtering again, washing with dry acetone, and drying over sulfuric acid. Yield: about 90 g.

PROPERTIES:

Blue-gray crystals, very deliquescent in air, soluble in water with a blue-violet color, readily soluble in alcohol, insoluble in acetone.

1349

REFERENCES:

- I. A. Hantzsch and E. Torke. Z. anorg. allg. Chem. 209, 72 (1932).
- II. A. Werner and A. Gubser. Ber. dtsch. chem. Ges. <u>34</u>, 1591 (1901).
- III. G. O. Higley. J. Amer. Chem. Soc. 26, 620 (1904).

#### Chloropentaaquochromium (III) Chloride

#### $[CrCl(OH_2)_5]Cl_2 \cdot H_2O$

- A)  $[CrCl_{2}(OH_{2})_{4}]Cl + H_{2}SO_{4} = [CrCl(OH_{2})_{5}]SO_{4} + 2 HCl$  $(\cdot 2 H_{2}O)$  $266.5 98.1 (3 H_{2}O)$ 327.7
- B)  $[CrCl(OH_2)_5]SO_4 + 2 HCl = [CrCl(OH_2)_5]Cl_2 + H_2SO_4$  $( \cdot 3 H_2O) ( \cdot H_2O)$ 327.7 266.5

A) PREPARATION OF  $[CrCl(OH_2)_5]SO_4 + 3 H_2O$ 

A solution of 26.8 g. of green chromium chloride hydrate  $[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O$  in an equal amount of water is allowed to stand for 24 hours at room temperature, and a mixture of 10 g. of conc.  $H_2SO_4$  and 4 g. of water is then added. The sulfate soon separates in bright green tablets.

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B) PREPARATION OF [C_rCl(OH_2)_5]Cl_2 \cdot H_2O
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A conc. aqueous solution of the sulfate, cooled to  $0^{\circ}$ C, is allowed to flow into ether at  $0^{\circ}$  while a stream of dry HCl is introduced. The yield is greater than 87%.

PROPERTIES:

Bright green, microcrystalline, very hygroscopic powder; readily soluble in water, alcohol and acetone. Differentiated from its isomers by its solubility in a mixture of equal volumes of ether and fuming hydrochloric acid. Insoluble in HCl-saturated ether.

REFERENCES:

- I. R. F. Weinland and Th. Schumann. Ber. dtsch. chem. Ges. <u>40</u>, 3094 (1907).
- II. M. Gutiérrez de Celis. An. Soc. Españ. Fisica Quim. <u>34</u>, 553 (1936), abstract in Chem. Zentr. <u>1936</u>, II, 1874.

### Hexaamminechromium (III) Chloride and Nitrate

#### [Cr(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>3</sub>, [Cr(NH<sub>3</sub>)<sub>6</sub>](NO<sub>3</sub>)<sub>3</sub>

# I. PREPARATION BY AUTOOXIDATION OF AN NH<sub>4</sub>Cl - CONTAINING AMMONIACAL SOLUTION OF A Cr (II) SALT

A solution of chromium (II) salt is prepared as indicated in the preparation of rhodochromium chloride (see p. 1359). This solution is forced under pressure (in the absence of air) into a flask containing a mixture of 525 g. of NH<sub>4</sub>Cl and 540 g. of ammonia (d 0.91). The vessel should be almost full at this point. The flask is stoppered at once with a cork which carries a gas outlet tube; the tube terminates under water. The flask is placed in cold water until H<sub>2</sub> evolution ceases (about 18-24 hours). The [Cr(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>3</sub> that deposits on the undissolved NH<sub>4</sub>Cl and that dissolved in the liquid are worked up separately.

The red solution is decanted and treated with an equal volume of 95% alcohol. The chloride, which settles after several hours, is washed by decantation with alcohol, filtered, rewashed with alcohol, and dried in air. It is then dissolved in lukewarm water and the solution passed through a filter into well-cooled nitric acid (d 1.39), whereupon the  $[Cr(NH_3)_6](NO_3)_3$  separates in long, yellow needles. The precipitate is washed several times by decantation with nitric acid, then with a mixture of 1 volume of nitric acid and 2 volumes of water, filtered, washed with alcohol until free of the acid, and dried in the air.

The product-containing  $NH_4Cl$  is treated several times with 150-ml. portions of water at room temperature, but only as long as the extracts are still yellow. They are treated with an equal volume of nitric acid (d 1.39; good cooling is essential). Yellow needles appear, either at once or after several hours; they are worked up as above. Total yield: 35-40 g.

The salt is purified by dissolving in a minimum quantity of cold water. The solution is passed through a filter into dilute nitric acid (1 vol. of nitric acid, d 1.4, and 2 vol. of water); the crystals are washed with alcohol and dried in air.

II.

$$CrCl_{3} + 6 NH_{3} = [Cr(NH_{3})_{6}]Cl_{3}$$

$$158.4 \quad 102.2 \quad 260.6$$

$$[Cr(NH_{3})_{6}]Cl_{3} + 3 HNO_{3} = [Cr(NH_{3})_{6}](NO_{3})_{3} + 3 HCl$$

$$260.6 \quad 189.0 \quad 840.2$$

The presence of NaNH<sub>2</sub> catalyst prevents the coproduction of  $[CrCl(NH_3)_5]Cl_2$ .

A hood with a good draft is needed; a one-liter Dewar flask is placed under this hood, charged with about 800 ml. of liquid NH<sub>3</sub>, and 0.5 g. of pure Na metal and 0.2 g. of  $Fe(NH_4)_2(SO_4)_2 \cdot 6 H_2O$ are added. (Instead of the Dewar flask, a one-liter beaker inserted in a second, 1.5-liter beaker may also be used.) After disappearance of the blue color of the NH<sub>3</sub> solution, 50 g. (nearly 0.3 mole) of  $CrCl_3$  is added with constant stirring (2-g. portions over a period of 1-2 hours). The brown precipitate is allowed to settle and the clear supernatant liquid is decanted or siphoned off.

The residue is transferred to a large dish and allowed to stand (with occasional stirring) until the odor of  $NH_3$  disappears and a lustrous yellow, free-flowing powder remains. The yield of crude  $[Cr(NH_3)_6]$  Cl<sub>3</sub> is almost quantitative (about 80 g.).

The crude product is purified by dissolving rapidly in a mixture of 10 ml. of conc. HCl and 150 ml. of water at 40 °C. After filtration, the solution is treated at once with 50 ml. of conc. nitric acid to precipitate pure  $[Cr(NH_3)_6](NO_3)_3$ . The liquid is allowed to cool to room temperature, the yellow crystalline salt is filtered on a Büchner funnel and washed with distilled water containing some HNO<sub>3</sub>, then with alcohol, and finally with ether. The product is dried in a vacuum desiccator in the absence of light and stored in a brown bottle. Yield: 80 g. (75%).

SYNONYM:

Luteochromic chloride or nitrate.

PROPERTIES:

The chloride (as well as the nitrate) forms orange-yellow crystals, only moderately soluble in water at room temperature (the nitrate in the ratio 1:40). Solubility is still further decreased by addition of nitric acid. All  $[Cr(NH_3)_6]^{3+}$  salts are sensitive to light even when dry. Decomposes slowly in solution, more rapidly on boiling, depositing chromium hydroxide. Heating with conc. HCl produces  $[CrCl(NH_3)_5]Cl_2$ .

REFERENCES:

- I. S. M. Jörgensen. J. prakt. Chem. <u>30</u>, 2 (1884).
- II. A. L. Oppegard and J. C. Bailar, Jr. in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 153.

## Chloropentaamminechromium (III) Chloride

## [CrCl(NH<sub>3</sub>)<sub>5</sub>]Cl<sub>2</sub>

## I. BY REACTION OF LIQUID NH<sub>3</sub> WITH CrCl<sub>3</sub>

Dry  $CrCl_3$  (8 g.) is added to liquid  $NH_3$ . The reaction starts at the boiling point of the  $NH_3$ , and the  $CrCl_3$  is transformed into a red

product. After evaporation of excess  $NH_3$ , the residue is triturated with 30 ml. of ice-cold water, filtered, then washed with some cold water until the filtrate is reddish. Concentrated nitric acid is added to the filtrate and  $[Cr(NH_3)_e](NO_3)_3$  is obtained (see p. 1351). Yield: about 7 g.

The red residue, consisting of  $[CrCl(NH_3)_5] Cl_2$ , is boiled with conc. HCl, cooled, mixed with water, filtered and washed with some cold water. It is then dissolved as rapidly as possible at 50°C in 400-500 ml. of water which is acidified with a few drops of H<sub>2</sub>SO<sub>4</sub>. The solution is immediately filtered through a large fluted filter paper and treated with an equal volume of conc. HCl. The salt precipitates in beautiful red crystals; after one hour, these are filtered, washed with 1:1 HCl, then with alcohol, and dried in a desiccator. Yield: about 5 g.

II. FROM THE RHODOCHLORIDE BY BOILING WITH HYDROCHLORIC ACID

The procedure for the preparation of rhodochromium chloride (see p. 1359) is followed, except that after the introduction of  $O_2$  the entire mixture is boiled for a few minutes with 2.5 times its volume of conc. HCl, whereupon the [CrCl(NH<sub>3</sub>)<sub>5</sub>] Cl<sub>2</sub> precipitates.

After cooling, the supernatant liquid is decanted. In 24 hours, additional purpureochromic chloride separates from the supernatant; it is, however, contaminated with  $NH_4Cl$ . The  $NH_4Cl$  is removed with dilute HCl; the residue is washed with alcohol and dried in a desiccator.

Yield: about 45 g. (from 60 g. of  $K_2Cr_2O_7$ ).

Purification is the same as in method I (solution in water containing some  $H_2SO_4$  and addition of HCl).

SYNONYM:

Purpureochromic chloride.

PROPERTIES:

Formula weight 243.54. Carmine-red crystals;  $d_4^{15.5}$  1.687. Solubility (16°C) 0.65 g./100 g. H<sub>2</sub>O. In aqueous solution, even on moderate heating, adds a water molecule to give  $[Cr(H_2O)(NH_3)_5]Cl_3$ . Space group  $V_h^{16}$ .

**REFERENCES**:

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- II. O. T. Christensen. J. prakt. Chem. 23, 57 (1881).

## Triethylenediaminechromium (III) Sulfate, Chloride and Thiocyanate

 $[Cr en_3]_2(SO_4)_3$ ,  $[Cr en_3]Cl_3 \cdot 3.5 H_2O$ ,  $[Cr en_3](SCN)_3 \cdot H_2O$ 

## A) ANHYDROUS CHROMIUM (III) SULFATE

 $Cr_2(SO_4)_3 \cdot 18 H_2O = Cr_2(SO_4)_3 + 18 H_2O$ 716.5 392.2

Heating of  $Cr_2(SO_4)_3 \cdot 18 H_2O$  for 2-3 days at 100-110°C gives a lumpy product; this is ground and dried further. Complete dehydration is indicated by the fact that the powder is no longer soluble in water.

## B) ANHYDROUS ETHYLENEDIAMINE

Since anhydrous ethylenediamine attacks cork and rubber stoppers, ground glass equipment must be used. Five hundred grams of NaOH and 875 ml. of commercial ethylenediamine hydrate are heated overnight on a steam bath. Two layers form; the upper layer is decanted, treated with additional 150 g. of NaOH for several hours; the supernatant is decanted again and distilled. B.p. 116-117°C at 760 mm. Yield: almost quantitative. (Propylenediamine can be dehydrated in the same manner.)

The ethylenediamine thus obtained still contains some water. Absolutely dry ethylenediamine reacts only very slowly with the  $Cr_2(SO_4)_3$ .

#### C) TRIETHYLENEDIAMINECHROMIUM (III) SULFATE

$$\frac{\operatorname{Cr}_2(\operatorname{SO}_4)_3}{392.2} + 6 \operatorname{H}_2 \operatorname{N} \cdot \operatorname{C}_2 \operatorname{H}_4 \cdot \operatorname{N} \operatorname{H}_2 = [\operatorname{Cr} \operatorname{en}_3]_2(\operatorname{SO}_4)_3}{360.6}$$

A 300-ml. Erlenmeyer flask, to which an air-cooled condenser is attached by a ground joint, is used to reflux 49 g. of  $Cr_2(SO_4)_3$ and 50 ml. of anhydrous ethylenediamine on a steam bath. Within one hour (and often much less), the sulfate begins to lose its bright green color and its powdery nature. If this should not occur after two hours, the reaction is induced by addition of a drop of water. From this time on, the flask must be shaken to and fro, to bring unreacted  $Cr_2(SO_4)_3$  into contact with the amine; the shaking is discontinued when a brown, solid mass forms; this is allowed to remain on the steam bath overnight. The solid, which is orangeyellow after cooling, is then broken up with a spatula, ground, washed with alcohol, and dried in air. Yield: 89 g. [95%, based on  $Cr_2(SO_4)_3$ ].

#### D) TRIETHYLENEDIAMINECHROMIUM (III) CHLORIDE HYDRATE

 $[Cr en_3]_2(SO_4)_3 + 6 HCl = 2[Cr en_3]Cl_3 + 3 H_2SO_4$  $(3.5 H_2O)$ 752.8 218.8 874.4

A solution of 32 g. of  $[Cr en_3]_2 (SO_4)_3$  in dilute HCl (5 ml. of conc. HCl and 30 ml. of water) is prepared at 60-65°C and rapidly filtered through a Büchner funnel. The filtrate is stirred and cooled in ice while 27 ml. of conc. HCl is added; the chloride  $[Cr en_3] Cl_3 \cdot 3.5 H_2O$  separates at once. Filtration yields 20 g. or 60% based on the sulfate used.

This chloride is still contaminated with sulfate. It may be purified by recrystallization from water. Thus 20 g. of the crude product is dissolved in 20 ml. of water at 65 °C. On cooling, 12 g. of pure chloride is obtained.

#### E) TRIETHYLENEDIAMINECHROMIUM (III) THIOCYANATE

$[Cr en_3]Cl_3$	+ 3 NH <sub>4</sub> SCN	$= [Cr en_3](SCN)_3$	+ 3 NH <sub>4</sub> Cl
(•3.5 H <sub>2</sub> O)		$(\cdot H_2O)$	
437.2	228.4	424,6	

A solution of 30 g. of  $[Cr en_3] Cl_3 \cdot 3.5 H_2O$  in 100 ml. of warm water is mixed, while ice-cooled and rapidly stirred, with a conc. aqueous solution of 36 g. of NH<sub>4</sub>SCN. The sparingly soluble  $[Cr en_3] (SCN)_3 \cdot H_2O$  separates at once. Filtration yields 30 g. of the crude product, or 94% based on the chloride charged.

For purification, the product is recrystallized from 100 ml. of water at 65°C, cooled, filtered, washed with alcohol and ether, and dried in air. Yield 23 g., or 77% based on the crude.

The bromide and the iodide can be obtained in exactly analogous fashion, that is, by addition of the corresponding ammonium salt.

Alternate methods: From violet chromium (III) chloride or dehydrated chrome alum, with ethylenediamine hydrate or ethylenediamine, respectively.

PROPERTIES:

The  $[Cr en_3]^{3+}$  salts are distinctly crystalline, orange-yellow substances, which are slightly sensitive to light even when dry. Their aqueous solutions have poor stability, particularly when heated or placed in sunlight: then the initial red color is followed shortly by complete decomposition. While the sulfate is extremely soluble in water and the chloride is also very soluble, the thiocyanate, the bromide and the iodide are relatively sparingly soluble. The chloride and the thiocyanate are readily converted by heating to the corresponding  $[CrX_2 en_2]^+$  salts (see the two preparations which follow).

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C. L. Rollinson and J. C. Bailar, Jr. in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 196. P. Pfeiffer. Ber. dtsch. chem. Ges. <u>37</u>, 4277 (1904); Reschke. Thesis, Univ. of Leipzig, 1925; M. Linhard and M. Weigel. Z. anorg. allg. Chem. <u>271</u>, 115 (1952).

## cis-Dichlorodiethylenediaminechromium (III) Chloride

## $[CrCl_2 en_2]Cl \cdot H_2O$

 $\begin{array}{ll} [\mathrm{Cr}\,\mathrm{en}_3]\mathrm{Cl}_3 &= [\mathrm{Cr}\mathrm{Cl}_2\,\mathrm{en}_2]\mathrm{Cl} \ + \ \mathrm{H}_2\mathrm{N}\cdot\mathrm{C}_2\mathrm{H}_4\cdot\mathrm{NH}_2 \\ (\cdot\,3.5\,\mathrm{H}_2\mathrm{O}) & (\cdot\,\mathrm{H}_2\mathrm{O}) \\ 437.2 & (296.7) \end{array}$ 

The [Cr en<sub>3</sub>] Cl<sub>3</sub>  $\cdot$  3.5 H<sub>2</sub>O, which serves as the starting material, is recrystallized from a 1% aqueous NH<sub>4</sub>Cl solution; this imparts a small NH<sub>4</sub>Cl content to the chloride complex, and the NH<sub>4</sub>Cl catalyzes the thermal decomposition. If the [Cr en<sub>3</sub>]Cl<sub>3</sub>  $\cdot$  3.5 H<sub>2</sub>O is prepared specifically as a starting material for this reaction, the NH<sub>4</sub>Cl may be added already during the recrystallization of the impure [Cr en<sub>3</sub>]Cl<sub>3</sub>  $\cdot$  3.5 H<sub>2</sub>O.

The recrystallized salt is dried and is then spread in a thin layer on a large watch glass, which is heated to 210°C. Careful control of the temperature is essential, since the rate of decomposition is too high above 215°C, while below 200°C the reaction is very slow. The evolution of ethylenediamine starts after a few minutes; the salt gradually becomes darker and after 1-2 hours turns red-violet. The course of the reaction is checked by the weight loss, which should approach the theoretical value of 30.6%.

A crude product, in satisfactory purity for many purposes, is obtained by washing with ice-cold conc. HCl. For further purification, it may be recrystallized as follows: The salt is dissolved rapidly in water at 70 °C, using 4 ml. of water per gram of salt, and the filtered solution is cooled in a cooling mixture. Then 1 ml. of ice-cold conc. HCl is added for each gram of the salt, whereupon small red-violet needles separate. These are filtered and washed with alcohol and ether. Yield: 0.45 g. (60%) per gram of starting compound.

Alternate method: From  $K_3[Cr(C_2O_4)_3]$  via two intermediate steps [A. Werner, Ber. dtsch. chem. Ges. <u>44</u>, 3135 (1911)].

PROPERTIES:

Small red-violet needles, readily soluble in water with a violet color. The solution becomes orange after a few hours, more rapidly when warmed.

**REFERENCES:** 

C. L. Rollinson and J. C. Bailar, Jr. in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 201; P. Pfeiffer. Ber. dtsch chem. Ges. <u>37</u>, 4277 (1904); M. Linhard and M. Weigel. Z. anorg. allg. Chem. <u>271</u>, 119 (1952).

## trans-Dithiocyanatodi(ethylenediamine)chromium (III) Thiocyanate

#### [Cr(SCN)<sub>2</sub> en<sub>2</sub>]SCN

trans-Dichlorodi(ethylenediamine)chromium (III) Chloride [CrCl<sub>2</sub> en<sub>2</sub>]Cl

a)

 $[Cr en_3](SCN)_3 = [Cr(SCN)_2 en_2]SCN + H_2N \cdot C_2H_4 \cdot NH_2$  $( \cdot H_2O)$ 424.6 346.6

The crude  $[Cr en_3] (SCN)_3 \cdot H_2O$  used as starting material is prepared according to the directions given on p. 1355 and recrystallized from a 1% NH<sub>4</sub>SCN solution as in the preparation of cis- $[CrCl_2 en_2] Cl \cdot H_2O$ . As in the latter case, the product is prepared by thermal decomposition, but at a temperature of 130°C (maximum 134°C). The theoretical weight loss is 18.40%. The product, which is a uniform yellow-red, is recrystallized several times from hot water, the solution concentration being such that the thiocyanate starts to crystallize slowly only after the solution is completely cold. This procedure yields 2 g. of pure  $[Cr(SCN)_2 en_2]SCN$  per 3 g. of crude; the material still contains 1-2 moles of water of hydration; this is removed in a desiccator.

b)

$$[Cr(SCN)_2 en_2]SCN \xrightarrow{Cl_2} [CrCl_2 en_2]Cl$$

$$346.6 \xrightarrow{278.7}$$

A fast stream of  $Cl_2$  is passed through an aqueous slurry of the thiocyanate obtained in (a); good cooling is necessary. The green crystalline powder which separates from the violet solution is essentially trans-dichlorosulfate and -chloride. About 0.6 g. of this
crude dichloro salt is obtained from 2 g. of thiocyanate. A concentrated solution of the crude salt in conc. HCl is placed in an  $H_2SO_4$  desiccator, which also contains a small dish with conc. HCl. The blue-green acid chloride [CrCl<sub>2</sub> en<sub>2</sub>] Cl · HCl · 2 H<sub>2</sub>O (0.3 g.) separates in one day. On heating to 100 °C, this is transformed into [CrCl<sub>2</sub> en<sub>2</sub>] Cl.

PROPERTIES:

 $Trans-[CrCl_2en_2]Cl$  consists of green crystals. A very thin layer of a conc. aqueous solution appears green, while thicker layers have a brown-red color.

REFERENCES:

- I. C. L. Rollinson and J. C. Bailar, Jr. in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 202; P. Pfeiffer. Z. anorg. Chem. <u>29</u>, 113 (1902).
- II. P. Pfeiffer. Ber. dtsch. chem. Ges. 37, 4282 (1904).

# Dichloroaquotriamminechromium (III) Chloride

# [CrCl<sub>2</sub>(OH<sub>2</sub>)(NH<sub>3</sub>)<sub>3</sub>]Cl

There are three position isomers: a, b, and c. The preparation starts from  $(NH_3)_3CrO_4$  and hydrochloric acid.

a) One gram of  $(NH_3)_3CrO_4$  (for preparation, see p. 1392) is carefully dissolved in 6 ml. of dilute HCl (the flask must be cooled with ice). It is added to the acid slowly in small portions, as soon as the vigorous reaction from the preceding portion has subsided. On addition of 10 ml. of conc. HCl and long standing in the cold, the red solution deposits red-violet snowflakelike crystals. These are recrystallized from the conc. aqueous solution by addition of conc. HCl. After washing with alcohol and ether, the crystals are dried over  $H_2SO_4$ . Yield: about 1.1 g.

b) About 2 g. of the chloride prepared in (a) is heated in HCl solution at about 60°C until the blue color of the solution is completely changed to green. By suction-filtration in a desiccator, dark green, needle-shaped crystals can be isolated from this solution; these are dried over  $H_2SO_4$ . This salt cannot be recrystallized; it is always contaminated with impurities.

c) If 8 ml. of conc. HCl is used to dissolve one gram of  $(NH_3)_3CrO_4$ under the same conditions as in (a), one obtains a bright green solution, from which gray, needle-shaped crystals soon separate. These are washed with alcohol and ether and dried over  $H_2SO_4$ . Yield: about 1 g.

PROPERTIES:

Formula weight 227.50. a) Red-violet dichroic crystals, soluble in water, giving a blue color. b) Dark green, needle-shaped crystals, soluble in water, giving a green color. c) Gray, needleshaped crystals, insoluble in cold water, soluble in warm  $H_2O$ , giving a red color.

REFERENCE:

E. H. Riesenfeld and F. Seemann. Ber. dtsch. chem. Ges. <u>42</u>, 422 (1909).

# Hexaureachromium (III) Chloride

$$[Cr(OCN_2H_4)_6]Cl_3 \cdot 3H_2O$$

 $[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O + 6 OCN_2H_4 =$ 

266.5

360,4

=  $[Cr(OCN_{2}H_{4})_{6}]Cl_{3} \cdot 3H_{2}O + 3H_{2}O$ 572.8

Green, crystalline chromium chloride hydrate  $[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O$  and somewhat more than the stoichiometric quantity of urea are dissolved in some water and treated with a few drops of HCl. The solution is concentrated in a drying oven at 75°C (or on the steam bath) until a crystalline crust forms. The crystal slurry thus obtained is dissolved in the minimum quantity of water at 50-60°C and rapidly filtered. The salt complex separates as green needles.

PROPERTIES:

Green needles, readily soluble in water, insoluble in absolute alcohol.

REFERENCE:

E. Wilke-Dörfurt and K. Niederer, Z. anorg. allg. Chem. <u>184</u>, 150 (1929).

# Rhodochromium Chloride

# [(NH<sub>3</sub>)<sub>5</sub>Cr(OH)Cr(NH<sub>3</sub>)<sub>5</sub>]Cl<sub>5</sub>

The preparation involves oxidation of an ammoniacal, NH<sub>4</sub>Clcontaining solution of Cr (II) salt.

Sixty grams of  $K_2Cr_2O_7$  powder is placed in a 2.5-liter beaker and covered with 200 ml. of conc. HCl and 75 ml. of alcohol (stirring). The resulting green solution of chromium (III) salt is reduced with zinc while still warm (but below 50°C) in the absence of air. The blue solution is poured into a mixture of 500 g. of NH<sub>4</sub>Cl and 750 ml. of conc. ammonia, the necessary good cooling being achieved by adding pieces of ice (or by immersion in ice). After decanting from undissolved NH<sub>4</sub>Cl, O<sub>2</sub> is passed through the liquid, which is shaken vigorously to achieve rapid oxidation. The liquid becomes red and rhodochloride deposits out abundantly. The salt is filtered, washed first with a mixture of 2 vol. of water and 1 vol. of conc. HCl. and then once with cold water. It is dissolved in cold water and the solution allowed to flow into a chilled mixture of 2 vol. of conc. HCl and 1 vol. of water, whereupon the rhodochloride reprecipitates almost completely. It is washes with 1:1 HCl, then with alcohol until free of acid, and dried in air in the dark. Yield: about 25 g.

PROPERTIES:

Formula weight 468.64. Pale crimson-red crystalline powder; contains 1 mole of  $H_2O$  when air-dried; this is slowly lost over conc.  $H_2SO_4$ .

**REFERENCES:** 

S. M. Jörgensen. J. prakt. Chem. <u>25</u>, 328 (1882); for composition, see K. A. Jensen. Z. anorg. allg. Chem. <u>232</u>, 257 (1937), as well as W. K. Wilmarth, H. Graff and S. T. Gustin. J. Amer. Chem. Soc. <u>78</u>, 2683 (1956).

# Erythrochromium Chloride

# [(NH<sub>3</sub>)<sub>5</sub>Cr(OH)Cr(NH<sub>3</sub>)<sub>4</sub>(OH<sub>2</sub>)]Cl<sub>5</sub>

Seven grams of rhodochromium chloride (preparation as above) is dissolved in 50 ml. of 2 N ammonia. This blue solution becomes pure crimson red in about 15 minutes; it is then cooled in ice and treated with 100 ml. of ice-cold, conc. HCl. The erythrochloride which precipitates is filtered, washed with some dilute HCl, then with alcohol and ether, and dried over  $H_2SO_4$ . Yield: 95%.

PROPERTIES:

Light-sensitive, crimson-red crystalline powder; more readily soluble in water than the rhodochloride.

**REFERENCES:** 

K. A. Jensen, Z. anorg. allg. Chem. <u>232</u>, 264 (1937). W. K. Wilmarth, H. Graff and S. T. Gustin, J. Amer. Chem. Soc. <u>78</u>, 2683 (1956).

Tris(2,2'-dipyridyl)chromium (II) Perchlorate

# [Cr(dipy)3](ClO4)2

 $Cr + 2 HCl = CrCl_{2} + H_{2}$ 52 (·aq.)  $CrCl_{2} + 3 dipy = [Cr(dipy)_{3}]Cl_{2}$ (·aq.)  $[Cr(dipy)_{3}]Cl_{2} + 2 NaClO_{4} = [Cr(dipy)_{3}](ClO_{4})_{2} + 2 NaCl$ (·aq.) 719

All operations are carried out under pure  $N_2$  and with deaerated liquids.

A solution of 0.26 g. of electrolytic Cr (preparation on p. 1335) in 2.5 ml. of 1:1 HCl is prepared. After the H<sub>2</sub> evolution ceases, the solution is diluted with 20 ml. of water, and 2.35 g. of 2,2'dipyridyl, dissolved in some methanol, is added. The solution, now a deep wine-red, is filtered through a fine fritted-glass filter. The filtrate is treated with a solution of 1 g. of NaClO<sub>4</sub> and 0.5 ml. of 70% HClO<sub>4</sub> in 50 ml. of water. A slurry of black-violet crystals is formed at once. This is filtered on fine fritted glass, washed with water, alcohol and ether, and dried in vacuum over  $P_2O_5$ . Yield: 3 g. (83% of theory).

#### PREPARATION OF 2,2-DIPYRIDYL

a) From FeCl<sub>3</sub> and pyridine in an autoclave [F. Hein and H. Schwedler, Ber. dtsch. chem. Ges. <u>68</u>, 681 (1935)]; b) refluxing of Raney nickel and pyridine [G. M. Badger and W. H. F. Sasse, J. Chem. Soc. (London) <u>1956</u>, 616].

The corresponding complexes with 1,10-phenanthroline  $[Cr \text{ phen}_3]_n X_n$  (where X = I or  $ClO_4$  and n = 1, 2, 3) can be prepared in a similar manner [S. Herzog, Chem. Techn. 8, 544 (1956)].

PROPERTIES:

Small black crystals; completely stable in air when dry, but oxidized when damp (acquires a yellow color). Sparingly soluble in water or methanol, giving an intense wine-red color. Insoluble in ether and benzene. In weak perchloric acid solution oxidized by air, forming yellow tris(2,2'-dipyridyl)chromium (III) perchlorate, which can be crystallized by concentrating the solution in the cold over  $H_2SO_4$ .

REFERENCES:

S. Herzog. Thesis, Univ. of Jena, 1952; F. Hein and S. Herzog. Z. anorg. allg. Chem. <u>267</u>, 337 (1952); G. A. Barbieri and A. Teitamanzi. Atti R. Accad. Lincei (Rome), Rend. [6] <u>15</u>, 877 (1932).

# Tris(2,2'-dipyridyl)chromium (I) Perchlorate [Cr(dipy)<sub>3</sub>]CIO<sub>4</sub>

 $2 [Cr(dipy)_{3}](ClO_{4})_{2} + Mg = 2 [Cr(dipy)_{3}]ClO_{4} + Mg(ClO_{4})_{2}$ 2.610

All operations are carried out under pure  $N_2$  and with air-free liquids.

Tris (2,2'-dipyridyl)chromium (II) perchlorate (1.0 g.; preparation as above) is covered with 250 ml. water, giving an opaque wine-red solution. This is treated with 60 mg. of Mg powder (about 3 times the stoichiometric quantity) and machine-shaken in a well-closed container. The solution becomes colorless after a maximum of 3 hours, and a fine, indigo-colored powder separates out. The powder, because of its fine particle size, imparts an apparent black-violet color to the solution on superficial examination. Now 3 g. of NH<sub>4</sub>ClO<sub>4</sub> is added and the mixture is shaken for an additional hour to dissolve the remaining Mg. After standing overnight, the dark-blue product is filtered through a fine fritted-glass filter, washed three times with 5-ml. portions of water, and dried in vacuum over P<sub>2</sub>O<sub>5</sub>. After a few hours, the preparation is dust-dry. Yield: about 0.65 g. (about 80% of theory).

PROPERTIES:

Indigo-blue powder; soluble in methanol, ethanol, acetone and pyriding, giving a deep, inky blue color; insoluble in water, benzene and ether. The solution is oxidized almost instantly in air, becoming lighter in color. The dry product reacts spontaneously with atmospheric  $O_2$  with considerable evolution of heat and loss of the 2,2'-dipyridyl.

**REFERENCES:** 

S. Herzog. Thesis, Univ. of Jena, 1952. F. Hein and S. Herzog. Z. anorg. allg. Chem. <u>267</u>, 337 (1952).

# Tris(2,2'-dipyridyl)chromium (0)

# [Cr(dipy )<sub>3</sub>]

 $Cr_2(CH_3COO)_4 + 3 \text{ dipy} = [Cr(dipy)_3] + Cr(III) \text{ and } Cr(II) \text{ complexes.}$  $(\cdot 2 H_2O)_{376,2} + 468.6 + 520.6$ 

Two grams of chromium (II) acetate hydrate is mixed (in the absence of air) with 2.49 g. of 2,2'-dipyridyl (equivalent to 1.5 moles of dipyridyl per g.-atom of Cr). Now, 40 ml. of deaerated water is added and the resulting suspension is machine-shaken for three hours. The black precipitate is filtered off from the deep-red mothor liquor (through a very fine fritted-glass filter), washed with water and alcohol, and dried over air-free  $P_2O_5$ . Yield: about 1 g.

Alternate methods: Reduction of tris(2,2'-dipyridyl)chromium (II) salt in tetrahydrofuran with sodium.

PROPERTIES:

Small black crystals; soluble in benzene, tetrahydrofuran, pyridine or dimethylformamide with a red color. Ignites in air with oxidation to  $Cr_2O_3$ .

REFERENCE:

S. Herzog, K. Chr. Renner and W. Schön. Z. Naturforsch. <u>12b</u>, 809 (1957).

#### Hexaphenylisonitrilochromium (0)

#### $[Cr(C_6H_5NC)_6]$

 $3 \operatorname{Cr}_2(\operatorname{CH}_3\operatorname{COO})_4 \cdot 2 \operatorname{H}_2O + \operatorname{C}_6\operatorname{H}_5\operatorname{NC} (\operatorname{excess}) \rightarrow$ 

1128.7

 $2 \left[ Cr(C_6H_5NC)_6 \right] + 4 Cr(III) \text{ complex}$ 1241.4

The reaction is carried out under  $N_2$ ; a large excess of phenyl isonitrile is desirable. Six grams of chromium (II) acetate  $Cr_2(CH_3COO)_4 \cdot 2H_2O$  is suspended in 40 ml. of methanol, and a solution of 20 g. of isonitrile in 10 ml. of methanol is added. After about one hour, well-formed garnet-red crystals separate from the deep black-red solution. The precipitate is filtered, washed with some methanol, and dried.

The yield is 5-6 g., which is almost quantitative, based on the disproportionation shown above.

PROPERTIES:

Garnet-red crystals with metallic, yellowish-green reflectance, stable in air, diamagnetic. M.p. 178.5°C (undergoes deformation at 151°C). Soluble in chloroform and benzene in the cold, readily soluble in the hot solvents. Can be recrystallized by reprecipitation with alcohol from a chloroform solution. Can be obtained from methylene chloride; in this case, large crystals, similar in appearance to pyrites, are obtained.

REFERENCE:

L. Malatesta, A. Sacco and S. Ghielmi. Gazz. Chim. Ital. <u>82</u>, 516 (1952).

If a Cr (II) halide is used instead of the Cr (II) acetate, the reaction is completely different: a crystalline precipitate of  $[Cr(RNC)_4Cl_2]$  (orange-red) or  $[Cr(RNC)_4Br_2]$  (olive brown), depending on the starting halide used, forms instantly. These compounds show a paramagnetism of 2.84 Bohr magnetons, corresponding to  $Cr^{2+}$ . They are again completely stable in air and can even be heated in water without decomposition; insoluble in ether, alcohol, benzene and carbon tetrachloride, but soluble in chloroform and methylene dichloride.

REFERENCE:

F. Hein and W. Kleinwächter. Private communication, unpublished.

# Chromium Orthophosphate

#### CrPO<sub>4</sub>

4 CrO <sub>3</sub>	-+-	$4 H_3PO_4$	+	$3 N_2 H_4 \cdot H_2 O$	=	4 CrPO <sub>4</sub>	+	$15 H_2O$	+	$3 N_2$
400.0		392.0		150,2		588,0		270.2		67.2 l.

A mixture of 11.6 g. of 85% H<sub>3</sub>PO<sub>4</sub> (d 1.69), 12.5 g. of CrO<sub>3</sub> (125% of the stoichiometric quantity) and 200 ml. of H<sub>2</sub>O is prepared, and 5.4 g. of 80% hydrazine hydrate (115% of the stoichiometric amount) in 100 ml. of H<sub>2</sub>O is added in drops and with stirring. After stirring for 15 minutes at 50°C, the precipitate is washed, suction-dried, and then dried for 2 hours at 100°C. Yield: 19 g. of amorphous

3.5-hydrate. Heating for 2 hours in vacuum at 800°C yields 13 g. of CrPO<sub>4</sub>, which gives a crystalline x-ray diffraction pattern.

PROPERTIES:

Hydrate: Turquoise green powder. d 2.15. Following the above directions gives a particle size of 0.1  $\mu$ , while a tenfold dilution of the reactants gives 1- $\mu$  particles.

Anhydrous: Gray-brown; insoluble in  $\rm H_2O$  and  $\rm CH_3COOH.$  d 3.05.

REFERENCES:

F. Wagenknecht. German Patents 1,046,597 (1957) and 1,056,104 (1957).

#### Chromium (II) Sulfate

CrSO<sub>4</sub> · 5 H<sub>2</sub>O

Twenty grams of coarse, very pure electrolytic chromium (>99.99% Cr) is placed in 150 ml. of  $H_2O$ , and 46 g. of conc.  $H_2SO_4$  is then added with agitation. The Cr is completely dissolved and massive crystals of  $CrSO_4 \cdot 5 H_2O$  precipitated from the deep blue, highly supersaturated solution. Concentration of the liquid in vacuum gives an almost theoretical yield of the product. The salt is filtered, washed with some ice-cold water, and dried in vacuum or in a stream of N<sub>2</sub>; any water present can be removed by thorough washing with acetone.

PROPERTIES:

Blue, massive crystals. Completely stable in air when dry. Solutions are instantly oxidized on contact with atmospheric  $O_2$ . Solubility (0°C): 21 g./100 g. H<sub>2</sub>O. Isotypic with CuSO<sub>4</sub> · 5 H<sub>2</sub>O.

**REFERENCE:** 

H. Lux and G. Illmann. Chem. Ber. <u>91</u>, 2143 (1958).

# Chromium (II) Salt Solutions

# I. PREPARATION BY ELECTROLYTIC REDUCTION

The electrolysis apparatus (see Fig. 318) comprises a 1.5-liter jar p and a cylindrical porous clay cell q of about 500-ml. capacity (height 17 cm., diameter 6.5 cm.); the cell is closed off with a

rubber stopper which carries a glass stirrer u with a mercury seal, a sampling tube s, a gas outlet tube t, and a lead cathode v having 230 cm.<sup>2</sup> of surface. The cell is surrounded by the Pb anode w.

The Pb cathode should be prepared according to directions given by Tafel (see the references below). It is suspended in 20% sulfuric acid and surrounded coaxially by a second cylindrical Pb electrode. The current (0.13 amp./in.<sup>3</sup>) is turned on, and the working electrode is operated first as an anode, then as a cathode (5 min.), and finally again as an anode (15 min.). After this, it is brown. It is washed with boiling water and dried.

The chromium (III) sulfate solution required for the electrolytic reduction is prepared as follows:  $SO_2$  is bubbled through a solution of 80 g. of  $K_2Cr_2O_7$ , 30 g. of conc.  $H_2SO_4$  and 450 g. of water until reduction is complete. Good cooling is necessary to prevent the transformation of violet to green



Fig. 318. Preparation of chromium (II) sulfate by electrolytic reduction. p jar; q porous clay cell; s sampling tube; t gas outlet tube; u stirrer with Hg seal; v lead cathode; w lead anode.

chromium (III) sulfate [the latter is not as readily reduced to chromium (II) sulfate]. The excess  $SO_2$  is driven off with a fast stream of air. The last traces of  $SO_2$  must be removed by brief boiling. The solution, whose volume is now about 50 ml., is transferred to the clay cell. The anodic electrolyte is 2 N H<sub>2</sub>SO<sub>4</sub>. Electrolysis proceeds at a current density of 0.13 amp./in.<sup>2</sup>, that is, at a current of 4.6 amp. The reduction takes 12 hours, but up to 24 hours may be required if a great deal of green chromium (III) sulfate is present. The course of the reduction can be followed by

removing samples and titrating with excess  $0.1 \text{ N KMnO}_4$ , adding KI, and back-titrating with  $0.1 \text{ N Na}_2\text{S}_2\text{O}_3$ .

#### **II.** PREPARATION BY REDUCTION WITH ZINC

It is best to use a zinc reductor. This consists of a vertical glass tube, 45 cm. long and 2 cm. I.D., with a glass stopcock at the lower end. It is two-thirds filled with zinc granules. Before use the contents of the column are amalgamated for 10 minutes with a 0.1 M HgCl<sub>2</sub> solution in 1 M HCl, then washed with a large quantity of water and finally with some 1 N H<sub>2</sub>SO<sub>4</sub>; during this operation the liquid level should always be above the zinc. The reduction proper is carried out by adding a solution of 90 g. of green chromium (III) chloride hydrate in 120 ml. of water and 30 ml. of 2 N H<sub>2</sub>SO<sub>4</sub> to the reductor tube; the rate of discharge from the reductor is so adjusted that only a pure, light blue chromium (II) salt solution drops into the directly attached storage or reaction vessel.

This solution obviously contains zinc salts. Solutions completely free of foreign salts are obtained either by dissolving chromium (III) acetate or, better, by dissolving electrolytic chromium in dilute HCl, as described, for example, in the procedure for tris(2,2'-dipyridyl)chromium (II) perchlorate.

#### APPLICATIONS

Useful for removing  $O_2$  from gases, for reductometric titration, and as a reductant in organic chemistry.

PROPERTIES:

Blue solution, very sensitive to air; storage stability is highly dependent on the purity of the starting materials.

**REFERENCES:** 

- I. Ch. W. Hofmann. Thesis, Univ. of Bern, 1947; R. Flatt and F. Sommer. Helv. Chim. Acta <u>25</u>, 684 (1942); A. Asmanow. Z. anorg. allg. Chem. <u>160</u>, 210 (1927); W. Traube and A. Goodson. Ber. dtsch. chem. Ges. <u>49</u>, 1679 (1916); J. Tafel. Z. phys. Chem. <u>34</u>, 187 (1900).
- II. E. Zintl and G. Rienäcker. Z. anorg. allg. Chem. <u>161</u>, 378 (1927); M. R. Hatfield in: L. F. Audrieth, Inorg. Syntheses, Vol III, New York-Toronto-London, 1950, p. 149.

# Chromium (II) Acetate Cr<sub>2</sub>(CH<sub>3</sub>COO)<sub>4</sub> · 2 H<sub>2</sub>O

I.

$$\begin{array}{c} CrCl_3 + \frac{1}{2}Zn = CrCl_2 + \frac{1}{2}ZnCl_2 \\ (\cdot 6 H_2O) \\ 266.5 \quad 32.7 \end{array}$$
(  $\cdot$  aq)

 $\begin{array}{c} \text{CrCl}_2 + 2 \text{ Na(CH_3COO)} = \frac{1}{2} \text{ Cr}_2(\text{CH}_3\text{COO)}_4 + 2 \text{ NaCl} \\ (\cdot \text{ aq}) & (\cdot 3 \text{ H}_2\text{O}) \\ & 272.2 & (\cdot 2 \text{ H}_2\text{O}) \\ & 188.1 \end{array}$ 

Pure chromium (II) acetate may be prepared only if oxygen is completely excluded. This condition is approached in the apparatus of Fig. 319.

Fig. 319. Preparation of chromium (II) acetate. *a* zinc reductor; *b* glass wool plug; *c* pinchcock or glass stopcock with 10-mm. bore; *d* reaction vessel; *e* fritted-glass funnel; *f* glass stirrer; *g* rubber cap to seal stirrer against outside air; *h* dropping funnel; *i* suction flask; *k* dropping funnel for washing liquids; *l* bubble trap for outgoing inert gas; *m* rubber sleeve for sealing large stopper.



The chromium (II) salt solution is obtained in a Zn reductor (see previous preparation). This consists of a glass tube a (45 cm. long and 2 cm. I.D.) in which a glass wool plug is inserted at b. The Zn granules filling the tube are amalgamated before use

(10 minutes with a 0.1 M HgCl<sub>2</sub> solution in 1 M HCl), then washed with large quantities of water and finally with some 1 N H<sub>2</sub>SO<sub>4</sub>; during this procedure the liquid level should always be above the zinc. A pinchcock or a glass stopcock with a 10-mm. bore is located at c. Reaction vessel d is attached with a rubber tube to the moderately coarse fritted-glass funnel e (diameter about 10 cm.). Glass stirrer f should provide thorough stirring of the precipitate during the washing and drying steps; it is held in place and turned by means of the rubber cap g which serves as a seal. Nitrogen or carbon dioxide (O<sub>2</sub>-free) is passed through the reaction vessel during the precipitation, and over the precipitate during the filtration. The gage pressure in the apparatus (governed by the liquid height in the trap l) should be as small as possible.

A solution of 90 g. of green chromium (III) chloride hydrate in 120 ml. of water and 30 ml. of 2 N  $H_2SO_4$  is poured into the reductor tube and its outflow rate so adjusted that only a pure light blue chromium (II) salt solution drops into the reaction vessel d. A filtered solution of 252 g. of Na acetate in 325 ml. of water is charged beforehand into the reaction vessel (via h). During the precipitation the vessel contents are stirred briefly by hand, using the stirrer provided.

After completion of the precipitation,  $N_2$  or  $CO_2$  is admitted into filter *e* and stopcock *c* is opened. In this operation the filtering flask *i* can be carefully put under a slight vacuum, provided a sufficient flow of  $N_2$  (or  $CO_2$ ) is maintained and the chromium (II) acetate is always surrounded only by the protective gas. The precipitate is washed on the filter with air-free distilled water, then several times with alcohol, and finally with peroxide-free ether, after which  $N_2$  or  $CO_2$  (H<sub>2</sub>O-free) is passed through for 24 hours. The chromium (II) acetate must be completely dry before it can be exposed to air, since it oxidizes at an appreciably faster rate when moist. Yield: 55 g.

The preparation can also be carried out with smaller quantities, e.g., one third of those given above. In this case, the dimensions are reduced to 7 cm. I.D. for the precipitating vessel d and the funnel e. The reductor need be only 35 cm. high (filled to 25 cm.). The drawing of Fig. 318 is based on dimensions appropriate to this case.

II. Cr + 2 HCl 
$$\rightarrow$$
 CrCl<sub>2</sub> · aq.  
52.0  
CrCl<sub>2</sub> · aq. + 2 CH<sub>3</sub>COONa  $\rightarrow \frac{1/2}{2}$  Cr<sub>2</sub>(CH<sub>3</sub>COO)<sub>4</sub> + 2 NaCl  
(· 2 H<sub>2</sub>O) 188.1

Two grams of electrolytic chromium is covered with a mixture of 6.2 ml. of conc. HCl and an equal volume of water (air should be excluded). After the start of the  $H_2$  evolution, about 10 additional ml. of water is added and the vessel is heated on a steam bath. When the evolution of  $H_2$  ceases, the sky-blue solution of chromium (II) chloride is slowly filtered through a fine fritted-glass funnel into a solution of 28 g. of sodium acetate in 40 ml. of deaerated water. The solution immediately turns red, and after a few seconds small glittering red crystals begin to precipitate. After overnight standing, these are filtered through fine fritted glass, washed five times with 10-ml. portions of water, dried with air-free CaCl<sub>2</sub>, and stored under N<sub>2</sub>. Yield: about 6 g.

Other means of obtaining chromium (II) salt solutions may be used instead of direct solution of the chromium used, provided the product solutions contain no foreign substances which would affect the precipitation of the acetate.

PROPERTIES:

Dark-red crystals, slightly soluble in water and alcohol. Insoluble in ether. When dry, stable in air for a few hours; stable indefinitely under N<sub>2</sub>. Drying over  $P_2O_5$  at 100 °C results in loss of the complexed water, change of color to brown, and increased sensitivity to air.

**REFERENCES:** 

S. Vanino. Handb. d. präp. Chemie [Handbook of Preparative Chemistry], Inorg. Section, Stuttgart, 1925, p. 710; E. Zintl and G. Rienäcker. Z. anorg. allg. Chem. <u>161</u>, 378 (1927);
M. R. Hatfield in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 149; K. H. Zapp. Unpublished, Freiburg i. Br.; S. Herzog. Unpublished, Jena; M. Kranz and A. Witkowska. Przemysl Chem. <u>37</u>, 470 (1958); Inorg. Syntheses, Vol. VI, 1960, p. 144.

# Chromium (II) Oxalate

#### CrC<sub>2</sub>O<sub>4</sub> · 2 H<sub>2</sub>O

$$\begin{array}{rrrr} CrSO_4 \cdot 5 H_2O &+ & Na_2C_2O_4 &= & CrC_2O_4 \cdot 2 H_2O &+ & Na_2SO_4 &+ & 3 H_2O \\ 238.2 & & 134.0 & 176.1 & 142.1 & 54.0 \end{array}$$

A dry mixture of 14 g. of  $CrSO_4 \cdot 5 H_2O$ , 8 g. of  $Na_2C_2O_4$ , and 0.25 g. of  $H_2C_2O_4 \cdot 2 H_2O$  is covered with about 150 ml. of  $O_2$ -free H<sub>2</sub>O under a protective blanket of  $N_2$ . This mixture is shaken vigorously. After some time,  $CrC_2O_4 \cdot 2 H_2O$  separates as a fine, crystalline, green precipitate. It is filtered, washed with cold  $H_2O$ , and dried over CaCl<sub>2</sub>, giving a yellowish-green powder. The yield is 80-85%.

Alternate method: Reaction of solid  $Na_2C_2O_4$  with concentrated solutions of chromium (II) salt obtained electrolytically (method of Walz).

REFERENCES:

H. Lux and G. Illmann. Chem. Ber. <u>91</u>, 2143 (1958); H. Walz. M.S. thesis, Univ. of Freiburg i. Br., 1958.

### Hexaaquochromium (III) Acetate

### [Cr(OH<sub>2</sub>)<sub>6</sub>](CH<sub>3</sub>COO)<sub>3</sub>

$$Cr(OH)_{3} \cdot 3 H_{2}O + 3 CH_{3}COOH = [Cr(H_{2}O)_{6}](CH_{3}COO)_{3}$$
  
157.1 180.2 337.2

I. Excess glacial acetic acid is added to the light blue-green Achromium (III) hydroxide (for preparation see p. 1345). The reaction proceeds with appreciable evolution of heat. The crystals (which precipitate after a few hours) are separated from the mother liquor, washed thoroughly with acetone and ether, and dried over  $H_2SO_4$ .

II. Alternate method: From chrome alum via the readily obtained dihydroxotetraaquochromium (III) sulfate.

PROPERTIES:

Needle-shaped blue-violet crystals, readily soluble in water; solvolyzed by alcohol.

REFERENCES:

I. A. Hantzsch and E. Torke. Z. anorg. allg. Chem. <u>209</u>, 78 (1932). II. A. Werner. Ber. dtsch. chem. Ges. 41, 3452 (1908).

# Dihydroxohexaacetatotrichromium (III) Acetate and Chloride

 $[Cr_3(OH)_2(CH_3COO)_6](CH_3COO) \cdot n H_2O, [Cr_3(OH)_2(CH_3COO)_6]C] \cdot 8 H_2O$ 

Prepared from  $CrO_3$ , glacial acetic acid and alcohol. a) A two-liter round-bottom flask fitted with a reflux condenser

is used. It is charged with 200 g. of  $CrO_3$  (sulfuric acid-free),

which is then covered with 400 ml. of commercial glacial acetic acid. The reaction is induced by careful heating on a water bath which is held below the boil. Since pure  $CrO_3$  does not react with very pure glacial acetic acid even at the boil, the reaction may be started by addition of some alcohol. If the reaction becomes too vigorous, the flask is cooled. When the evolution of  $CO_2$  begins to subside, the flask contents are refluxed for about 2 hours on a rapidly boiling water (or steam) bath. The thick, brown contents of the flask, which consist of hexaacetatotrichromium chromates, are allowed to cool somewhat. To complete the reduction of any chromic acid still present, first 50% alcohol and then 96% alcohol (about 100 ml. of alcohol in all) is added in small portions through the condenser. The flask is now heated for one hour on a steam bath, and the green liquid is then concentrated on a water bath. The green diacetate powder has the formula

 $[Cr_3OH(H_2O)(CH_3COO)_6](CH_3COO)_2 \cdot H_2O.$ 

The monoacetate hexahydrate is obtained by dissolving the powder in some water and allowing evaporation to take place over  $H_2SO_4$ . The monoacetate tetrahydrate crystallizes in long prisms when an aqueous solution of the diacetate is treated with acetone.

PROPERTIES:

Formula weight 675.41 (4  $H_2O$ ), 711.45 (6  $H_2O$ ). Green, water-soluble crystals.

b) Evaporation of a solution of the diacetate in dilute HCl over  $H_2SO_4$  yields  $[Cr_3(OH)_2(CH_3COO)_6]$  Cl · 8  $H_2O_2$ .

PROPERTIES:

Formula weight 723.90. Dark green prisms, may be recrystallized from water.

REFERENCES:

R. F. Weinland and E. Büttner. Z. anorg. allg. Chem. 75, 329, Anm. 1 (1912); R. Weinland and P. Dinkelacker. Ber. dtsch. chem. Ges. <u>42</u>, 3010, 3012 (1909).

# Potassium Trioxalatochromate (III)

#### $\mathbf{K}_{3}\left[\mathbf{Cr}(\mathbf{C}_{2}\mathbf{O}_{4})_{3}\right]\cdot\mathbf{3}\mathbf{H}_{2}\mathbf{O}$

Prepared by treatment of oxalic acid and potassium oxalate with  $K_2Cr_2O_7$ .

A concentrated aqueous solution containing 12 g.  $K_2Cr_2O_7$  is added dropwise with stirring to a solution containing 27 g. of oxalic acid dihydrate and 12 g. of neutral potassium oxalate monohydrate. The mixture is then evaporated to a small volume and slowly cooled to bring about crystallization.

SYNONYM:

Potassium chromium oxalate.

PROPERTIES:

Formula weight 485.4. Black-green, monoclinic scales with transparent blue edges. Readily soluble in water.

REFERENCE:

H. Hecht. Präparative anorg. Chemie [Preparative Inorganic Chemistry], Berlin-Göttingen-Heidelberg, 1951, p. 158.

### Potassium Hexacyanochromate (III)

#### $K_3[Cr(CN)_6]$

#### $Cr(CH_3COO)_3 + 6 KCN \rightarrow K_3[Cr(CN)_6] + 3 CH_3COOK$

Seventeen grams of  $CrO_3$  or 25 g. of  $K_2Cr_2O_7$  is treated with 70 ml. of HCl (45 ml. of conc. HCl + 25 ml. of water) and reduced while hot by addition of a total of 25 ml. of ethanol in small portions, A very slight excess of ammonia is added to the boiling solution. The precipitate of  $Cr(OH)_3$  is filtered hot through a fluted filter paper, washed several times with hot water, and finally dissolved in some dilute acetic acid. This solution is evaporated almost to dryness in order to remove the excess acetic acid. The residue is taken up in 150 ml. of water, filtered, and poured into a boiling solution of 100 g. of KCN in 200 ml. of water (use a hood!). The very dark-red solution thus formed is evaporated on a steam bath. A brownish-black solid usually separates; this is removed by filtration. On further concentration, bright-yellow crystals deposit on the walls. The mother liquor also yields additional fractions. The product is recrystallized two or three times from water and dried over H<sub>2</sub>SO<sub>4</sub>. The yield is 38 g. (70%).

#### PROPERTIES:

Formula weight 325.41. d 1.71. Bright yellow, monoclinic crystals, isomorphous with  $K_3[Fe(CN)_6]$ ; decomposes above 150°C. Solubility at 20°C: 30.96 g./100 g. water; insoluble in alcohol.

Aqueous solutions tend to decompose, especially in light or on heating, separating  $Cr(OH)_3$ .

REFERENCES:

F. V. D. Cruser and E. H. Miller. J. Amer. Chem. Soc. <u>28</u>, 1133 (1906); O. T. Christensen. J. prakt. Chem. [2] <u>31</u>, 163 (1885); see also J. H. Bigelow in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 203.

### Potassium Hexathiocyanatochromate (II)

#### $K_3[Cr(SCN)_6] \cdot 4 H_2O$

 $\begin{array}{rl} KCr(SO_4)_2 + \ 6 \ KSCN = K_3[Cr(SCN)_6] + 2 \ K_2SO_4 \\ & \begin{array}{r} 12 \ H_2O \\ & \begin{array}{r} 499.4 \end{array} & \begin{array}{r} 583.0 \end{array} & \begin{array}{r} 589.8 \end{array} \end{array}$ 

A moderately concentrated aqueous solution of 6 parts of KSCN and 5 parts of chrome alum is heated for 2 hours on a steam bath, and is then concentrated in a dish until the cooled residual liquid solidifies to a mass of red crystals. This solid is extracted with absolute alcohol, in which the  $K_3[Cr(SCN)_6]$  dissolves very readily while  $K_2SO_4$  remains as a residue. After evaporation of the filtered alcohol extract, the salt is recrystallized once more from alcohol.

The analogous ammonium salt  $(NH_4)_3[Cr(SCN)_6] \cdot 4 H_2O$  is obtained in the same manner, except that reaction in the solution of  $NH_4SCN$  and chromium ammonium alum takes place only after a brief period of boiling.

SYNONYM:

Potassium chromium thiocyanate.

PROPERTIES:

Lustrous crystals; dark red-violet in reflected light and garnet red in transmitted light. The salt remains unchanged in air or over  $H_2SO_4$ ; it loses its water of crystallization only when heated to 110°C. One part dissolves in 0.72 parts of water and in 0.94 parts of alcohol; d<sup>16</sup> 1.711.

REFERENCE:

J. Roesler. Liebigs Ann. <u>141</u>, 185 (1867).

# Trilithium Hexaphenylchromate (III)

# $Li_3Cr(C_6H_5)_6 \cdot 2.5(C_2H_5)_2O$

 $\begin{array}{rl} & {}^{ether} \\ CrCl_3 \ + \ 6 \operatorname{Li}(C_6H_5) \rightarrow \\ 158.4 & 504.2 \end{array} \rightarrow \begin{array}{rl} Li_3Cr(C_6H_5)_6 \cdot 2.5 \ (C_2H_5)_2O \ + \ 3 \ LiCl \\ 127.3 \end{array}$ 

All operations are conducted in the absence of air and moisture, using high-purity  $N_2$  as a protective gas.

a) The required solution of phenyllithium is prepared in a 500-ml. three-neck flask fitted with a reflux condenser and Hg seal, a highspeed Hg-sealed mechanical stirrer, a gas inlet tube, and a dropping funnel. The procedure is as follows: Clean, finely cut lithium (6 g.) is covered with 100 ml. of ether that has been freshly distilled over benzophenonesodium (referred to hereafter as ketyl ether). With the stirrer operating at high speed, 64 g. of freshly distilled bromobenzene, dissolved in 200 ml. of ketyl ether, is added dropwise at a rate sufficiently fast to keep the reaction solution boiling vigorously. After all the bromobenzene has been added, the solution is refluxed for one hour. After cooling, it is filtered through fine fritted glass. The clear solution thus obtained has a phenyllithium concentration of about 10%.

b) The preparation of the lithium chromium phenyl complex employs the same apparatus as described in (a). However, the dropping funnel is replaced by a tap-injection bulb containing 10 g. of anhydrous, very finely powdered chromium (III) chloride. With vigorous, high-speed stirring of the lithium phenyl solution in the flask, the chromium chloride is slowly introduced by tapping the bulb. The course of the reaction is monitored by observing the decrease in the number of black particles of chromium chloride. After 10-12 hours, the reaction is discontinued without waiting for complete conversion of the solid chromium chloride. The nascent yellow precipitate is filtered through fine fritted glass. By cooling the black-brown filtrate to  $-10^{\circ}$ C, a portion of the complex is obtained in beautiful crystals. The reaction residue is rinsed back into the three-neck flask with 200 ml. of ketyl ether and again collected on the fritted glass.

The reaction flask is now replaced with a reflux condenser which is attached to the  $N_2$ -generating apparatus (to equalize the pressure). The receiver flask is heated and the ether is distilled through the fritted glass plate and onto the residue; by cooling the receiver flask, the ether is suction-drawn through the residue back into the flask. This operation is repeated until the residue is colorless. On cooling, most of the complex compound crystallizes in the receiver. The crystal slurry thus obtained is recrystallized from a large quantity of ketyl ether; or it is extracted with fresh ketyl ether as described above. The mother liquor must be yellow-brown and free of halogens. Yield: about 15 g.

PROPERTIES:

Yellow-orange crystals; soluble in ether, benzene and tetrahydrofuran; sensitive to air and moisture; completely hydrolyzed by water or alcohol.

**REFERENCE:** 

F. Hein and R. Weiss. Z. anorg. allg. Chem. 295, 145 (1958).

# Ammonium Tetrathiocyanatodiamminechromate (III)

 $NH_4[Cr(SCN)_4(NH_3)_2] \cdot H_2O$ 

Prepared by fusion of  $NH_4SCN$  with  $(NH_4)_2Cr_2O_7$  and extraction with water.

An enamel cooking pot of at least 4-liter capacity is charged with 800 g. (10.5 moles) of  $NH_4SCN$  and carefully heated; several small flames are used to provide as uniform heating as possible. The mass is stirred with a glass test tube which contains a thermometer; the heating is continued until the solid is partly melted and the temperature is 145-150 °C. Now, an intimate mixture of 170 g. (0.675 mole) of finely powdered  $(NH_4)_2Cr_3O_7$  and 200 g. (2.6 moles) of NH<sub>4</sub>SCN is added in portions of 10-12 g. with continuous stirring. A fairly vigorous reaction begins after 10 such portions have been added;  $NH_3$  is evolved and the temperature rises to 160 °C. The flames are now extinguished and the rest of the mixture is added to the melt in such a way as to maintain the temperature at 160 °C.

Stirring is continued as the melt cools; the solid product which deposits on the walls of the vessel is scraped away, ground to a fine powder while still warm, and stirred in a large beaker with 750 ml. of ice water. After 15 minutes, the insoluble residue is freed of mother liquor as completely as possible (suction-filtration, no washing). It is then stirred into 2.5 liters of water, preheated to  $65^{\circ}$ C. The temperature is rapidly restored to  $60^{\circ}$ C and the solution is filtered all at once through a funnel heated with hot water (heating above  $65^{\circ}$  causes rapid decomposition, with production of a blue color and generation of HCN).

The hot filtrate is placed overnight in an ice chest, and the separated crystals are then filtered with suction. The mother liquor is used for another extraction of the residue at 60°C, thus affording an additional quantity of crystalline Reinecke salt.

Finally, 12-13 additional grams of product may be obtained by concentrating the mother liquor to 250-300 ml. under reduced pressure at 40-50 °C.

The total yield of air-dry Reinecke salt amounts to 250-275 g. (52-57% of theory).

The insoluble residue from the second extraction (about 130-135 g.) is composed predominantly of Morland salt, i.e., guanidium tetrathiocyanatodiamminechromate (III).

USES:

Used for the isolation of amines, amino acids, complex cations and organometallic bases; it forms sparingly soluble salts with all of the above; these salts usually crystallize well. Also used as reagent in quantitative determination of Cu and Hg (procedure of C. Mahr) and of quaternary onium cations (procedure of F. Hein).

SYNONYM:

Reinecke salt.

PROPERTIES:

Formula weight 354.45. Ruby-red, lustrous, light-sensitive leaflets, which lose their water of crystallization on drying at 100°C and form scarlet cubes and rhombododecahedra. Both forms are readily soluble in cold water, alcohol, acetone and moist ethyl acetate, insoluble in benzene. Decomposed by boiling water.

REFERENCES:

H. D. Dakin. Org. Syntheses 15, 74 (1935); Coll. Vol. II, 555 (1943).

#### Tetrathiocyanatodiamminechromic (III) Acid

### $H[Cr(SCN)_4(NH_3)_2]$

 $\begin{array}{rrrr} NH_4[Cr(SCN)_4(NH_3)_2] &+ & HCl = H[Cr(SCN)_4(NH_3)_2] &+ & NH_4Cl \\ (\cdot H_2O) & & & (\cdot 2 H_2O) \\ 354.5 & & 36.5 & & 355.4 \end{array}$ 

A concentrated aqueous solution of  $\rm NH_4[Cr(SCN)_4 (NH_3)_2] \cdot H_2O$ (see preceding preparation) is treated with a small excess of hydrochloric acid, then extracted thoroughly with ether. The free acid is absorbed in the ether with an intense dark red color; addition of NaCl makes the extraction almost quantitative. Evaporation of the ethereal solution in vacuum over  $\rm H_2SO_4$  and KOH yields a red mass which loses its solubility in ether after standing for a few

days. The product is recrystallized from 50°C water, in which it dissolves very readily, except for a small yellow residue. On cooling, small red scales separate; these are again recrystallized from water.

SYNONYM:

Reinecke acid.

PROPERTIES:

Lustrous red leaflets; readily soluble in water, alcohol and acetone. Heating for several days at 70°C renders the acid anhydrous; further heating at 110-115°C imparts a darker color. The undried compound decomposes between 80 and 90°C, puffing up and evolving water.

**REFERENCE:** 

R. Escales and H. Ehrensperger. Ber. dtsch. chem. Ges. <u>36</u>, 2681 (1903).

# Ammonium Tetrathiocyanatodianilinochromate (III)

$$NH_4[Cr(SCN)_4(C_6H_5NH_2)_2] \cdot 1^{1/2} H_2O$$

a)  $\begin{array}{rcl} & & & \text{KCr}(\text{SO}_4)_2 \ + \ 6 \ \text{KSCN} \ = \ \text{K}_3[\text{Cr}(\text{SCN})_6] \ + \ 2 \ \text{K}_2 \text{SO}_4 \\ & & (\cdot 12 \ \text{H}_2 \text{O}) \\ & & 499.4 & 583.0 & 517.8 \end{array}$  $\begin{array}{rcl} & & \text{K}_3[\text{Cr}(\text{SCN})_6] \ + \ 4 \ \text{C}_8 \ \text{H}_5 \text{NH}_2 \ + \ \text{CH}_3 \text{COOH} \\ & & 517.8 & 372.5 & 60.1 \\ & & = \ (\text{C}_6 \ \text{H}_5 \text{NH}_2)_2 \ \text{H}[\text{Cr}(\text{SCN})_4(\text{C}_6 \ \text{H}_5 \text{NH}_2)_2] \ + \ 2 \ \text{KSCN} \ + \ \text{CH}_3 \text{COOK} \\ & & 657.8 \end{array}$ 

b) 
$$(C_{6}H_{5}NH_{2})_{2}H[Cr(SCN)_{4}(C_{6}H_{5}NH_{2})_{2}] + NH_{3} = \frac{657.8 17.0}{NH_{4}[Cr(SCN)_{4}(C_{6}H_{5}NH_{2})_{2}] + 2C_{6}H_{5}NH_{2}}{458.6 186.2}$$

a) A mixture of 500 g. of chrome alum, 600 g. of KSCN, and 500 ml. of water is heated for 4 hours on a steam bath. The solution is cooled, 500 ml. of aniline is added, and the mixture is stirred for 3 hours at  $60^{\circ}$ C on a water bath. It is then again cooled and a mixture of 6 liters of water and 600 ml. of glacial acetic acid is added. After a few hours the precipitate is filtered and dissolved

in 1.5-2 liters of cold methanol. This solution is filtered, and 6 liters of water is added, whereupon  $(C_{eH_5}NH_2)_2H[Cr(SCN)_4-(C_{eH_5}NH_2)_2]$  precipitates as a thick, violet crystal slurry. After a further precipitation from methanol-water, the yield is 330 g. b) Four hundred grams of this anomalous anilinium salt is treated with 600 ml. of methanol and 300 ml. of conc. ammonia. This solution is cooled in ice and 3 liters of water is slowly added; the crude ammonium thiocyanoto-aniline complex precipitates. After filtering with suction, it is treated once more in the same manner with methanol, ammonia and water. Yield: about 200 g.

USE:

Separation of amino acids, especially proline.

SYNONYMS:

Ammonium salt of rhodanilic acid; ammonium rhodanilate.

PROPERTIES:

Violet-red crystals, somewhat soluble in water, very soluble in methanol, acetone and ethyl acetate. The solutions decompose on boiling. Insoluble in ether, benzene and chloroform.

**REFERENCE:** 

M. Bergmann. J. Biol. Chem. 110, 476 (1935).

# Potassium Tetrathiocyanatodipyridinochromate (III)

#### $K[Cr(SCN)_4py_2] \cdot 2H_2O$

Prepared from  $K_{3}[Cr(SCN)_{6}]$  and pyridine.

Ten parts of  $K_3[Cr(SCN)_6]$  (for preparation, see p. 1374), dried at 110°C, is heated with 30 parts of anhydrous pyridine in a small flask (4 hours on the water bath, in the absence of moisture). The hot solution is then poured into a crystallizing dish and allowed to chill in an ice chest. The solid which crystallizes is a mixture of KSCN,  $py_3 \cdot H[Cr(SCH)_4py_2]$ , and  $[Kpy_4][Cr(SCN)_4py_2]$ . It is suction-dried and placed on a clay plate. The complex potassium salt deliquesces over a period of 1-2 days, and the KSCN is extracted with water at room temperature, while the  $K[Cr(SCN)_4py_2]$ is extracted with hot water. When cooled, the resulting red solutions gradually deposit small, lustrous red crystals of the potassium salt. The residue remaining after the hot water extraction affords pure dipyridinium salt  $py_2 \cdot H[Cr(SCN)_4py_2]$ . PROPERTIES:

Formula weight 517.66. Small red crystals, which become anhydrous on heating to 110 °C. Almost insoluble in cold water, but somewhat soluble in warm H<sub>2</sub>O. Completely insoluble in benzene, chloroform and ether; very soluble in aqueous and absolute ethyl alcohol, methanol, ethyl acetate and pyridine; very readily soluble in acetone.

REFERENCE:

P. Pfeiffer. Ber. dtsch. chem. Ges. 39, 2121, 2123 (1906).

# Trichlorotriaquochromium

 $[CrCl_3(OH_2)_3]$ 

 $3[CrCl_{2}(OH_{2})_{4}]Cl = 2[CrCl_{3}(OH_{2})_{3}] + [CrCl_{2}(OH_{2})_{4}]Cl \cdot 2 H_{2}O$ 691.3 424.9 266.5

Green chromium chloride hydrate  $[CrCl_2(OH_2)_4]Cl \cdot 2 H_2O$  is converted into  $[CrCl_2(OH_2)_4]Cl$  on standing for 3 days in a vacuum desiccator over conc.  $H_2SO_4$ . It is then suspended in absolute ether;  $[CrCl_3(OH_2)_3]$  is formed by disproportionation and dissolves with a brown-violet color. On evaporation of the ethereal solution in the absence of atmospheric moisture,  $[CrCl_3(OH_2)_3]$  is obtained as an amorphous brown powder.

PROPERTIES:

Formula weight 212.43. Brown, amorphous, very hygroscopic powder, rapidly altered by traces of water. Soluble in water with a yellow-green color, which quickly becomes pure green owing to a reaction. Solutions in ether may be stored without change if moisture is absent.

REFERENCES:

A. Recoura. Comptes Rendus Hebd. Séances Acad. Sci. <u>194</u>, 229 (1932); <u>196</u>, 1854 (1933); see also F. Hein. J. prakt. Chem. <u>153</u>, 168 (1939).

# Trichlorotriethanolochromium

 $[CrCl_3(C_2H_5OH)_3]$ 

 $CrCl_3 + 3 C_2H_5OH = [CrCl_3(C_2H_5OH)_3]$ 158.4 138.2 296.6

Dried  $CrCl_3$  is refluxed (in the absence of moisture) with absolute alcohol and a small piece of zinc (or  $CrCl_2$ ). The  $CrCl_3$ 

dissolves; the solution, which is red when hot and green when cold, is concentrated in a vacuum desiccator over conc.  $H_2SO_4$ . The red crystals which deposit are washed with some absolute alcohol and ether, and stored dry.

PROPERTIES:

Dark red, hygroscopic crystals; soluble in alcohol, acetone and chloroform with a red color which soon becomes green. The aqueous solution decomposes rapidly.

**REFERENCE:** 

I. Koppel. Z. anorg. Chem. 28, 471 (1901).

#### Trichlorotriamminechromium

#### $[CrCl_3(NH_3)_3]$

Prepared from  $(NH_3)_3CrO_4$  and hydrochloric acid.

Five grams of triamminechromium tetroxide (for preparation, see p. 1392) is introduced into 50 ml. of well-cooled conc. HCl (constant stirring). The resulting gray- to blue-green precipitate is filtered off. The neutral complex, which deposits from the filtrate after standing for 1-2 days, is filtered with suction and washed with water until the washings become colorless. It is then dried by washing with alcohol and ether.

PROPERTIES:

Formula weight 209.48. Blue crystals with greenish tinge, insoluble in cold  $H_2O$ . Dissolution in warm  $H_2O$  causes aquation to  $[CrCl_2(OH_2)(NH_3)_3]Cl$ . Presumably trans form.

REFERENCE:

A. Werner. Ber. dtsch. chem. Ges. 43, 2289 (1910).

# Trichlorotripyridinechromium

#### [CrCl<sub>3</sub>py<sub>3</sub>]

I.

$$CrCl_3 + 3 py = [CrCl_3py_3]$$
  
158.4 237.3 395.7

The  $CrCl_3$ , in excess of dry pyridine, is refluxed in the presence of a small granule of  $CrCl_2$ . The  $CrCl_3$  dissolves completely after some time, giving a green color. The solution is filtered and cooled, whereupon the  $[CrCl_3py_3]$  crystallizes out. On distilling the pyridine from the mother liquor, the compound can be obtained in almost quantitative yield.

II. Addition of  $H_2O$  to a pyridine solution of green chromium chloride hydrate  $[CrCl_2(H_2O)_4]Cl \cdot 2 H_2O$  yields a green powder which consists essentially of a mixture of  $[CrCl_3py_3]$  and  $[Cr(OH)_2(H_2O)_2py_2]Cl$ . When this mixture is treated with HCl, the latter salt goes into solution as  $[Cr(H_2O)_4py_2]Cl_3$ , giving a deep red color. The residue consists of crude  $[CrCl_3py_3]$ . This is dissolved in conc. HCl and reprecipitated by pouring the filtered solution into a large amount of water. Finally the  $[CrCl_3py_3]$  is recrystallized once more from pyridine.

PROPERTIES:

Green leaflets, readily soluble in pyridine, chloroform, acetone and conc. HCl; sparingly soluble in ethyl alcohol; insoluble in water, ether, benzene and naphtha.

REFERENCES:

P. Pfeiffer. Z. anorg. Chem. 24, 282 (1900); 55, 99 (1907).

# Chromium (III) Glycinate

# (H<sub>2</sub>NCH<sub>2</sub>COO)<sub>3</sub>Cr

 $CrCl_3 \cdot 6 H_2O + 3 H_2NCH_2COOH + 3 NaOH$ 266.5 225.2 120.0 $= (H_2NCH_2COO)_3Cr + 3 NaCl + 3 H_2O$ 274.2

An aqueous solution of one mole of green chromium chloride hydrate and 3 moles of glycine is boiled while 3 moles of NaOH is added gradually. This gives a dark-red solution from which a violet compound separates. The latter is filtered off while the mixture is still hot. The filtrate, after cooling and standing in vacuum over  $H_2SO_4$ , deposits still more of the violet compound, together with larger red crystals. After suction-filtration and drying, the heavy red crystals are separated from the lighter violet ones by slurrying with alcohol. In this way, both compounds are obtained in analytically pure state.

PROPERTIES:

Red crystals = chromium (III) glycinate,  $(H_2NCH_2COO)_3Cr$ . Violet crystals = so-called ''basic'' chromium (III) glycinate,  $(NH_2CH_2COO)_2Cr(OH)_2Cr(OOCCH_2NH_2)_2 \cdot H_2O$ .

Both compounds are sparingly soluble in water and insoluble in organic solvents.

Chromium (III)  $\alpha$ -alaninate can be obtained in an analogous manner. If the reaction is allowed to take place in conc. solution, the red chromium (III) alaninate separates; the "basic" chromium (III) alaninate is obtained by evaporation of the solution.

REFERENCE:

H. Ley. Ber. dtsch. chem. Ges. 45, 380 (1912).

# Chromium (III) Xanthate

#### $[(C_2H_5OCS_2)_3Cr]$

 $\begin{array}{rcl} \text{KCr}(\text{SO}_4)_2 &+& 3\,\text{C}_2\text{H}_5\text{OCS}_2\text{K} &=& [(\text{C}_2\text{H}_5\text{OCS}_2)_3\text{Cr}] &+& 2\,\text{K}_2\text{SO}_4 \\ (\cdot\,12\,\text{H}_2\text{O}) && 480.9 && 415.6 \\ && 499.4 \end{array}$ 

A solution of 20 g. of potassium xanthate in some water is treated with a solution of 23 g. of chrome alum. The blue-black compound which precipitates is filtered off with suction and dried on a clay plate. It is dissolved in pyridine, and water is added in drops until a permanent clouding is obtained. The solution is then allowed to stand undisturbed to bring about crystallization. The crystals are separated by filtration and dried in vacuum.

PROPERTIES:

Dark-blue crystalline powder, soluble in organic media, insoluble in water.

REFERENCE:

J. V. Dubsky. J. prakt. Chem. <u>90</u>, 118 (1914).

# Chromium (III) Acetylacetonate (C<sub>5</sub>H<sub>7</sub>O<sub>2</sub>)<sub>3</sub>Cr

L  $Cr(CH_3COO)_3 + 3C_5H_8O_2 = [(C_5H_7O_2)_3Cr] + 3CH_3COOH$  $(\cdot 6H_2O)_{337.2} 300.3 349.3$ 

A mixture of 40 g. of  $[Cr(OH_2)_6](CH_3COO)_3$  (for preparation, see p. 1371), 150 ml. of water, 40 g. of acetylacetone and 50 ml. of

2 N acetic acid is heated until solution is complete and crystallization of the internal complex begins. Then the solution is boiled for a short time until the liquid bumps vigorously. It is cooled gradually, then chilled in ice and filtered. The first crop affords 18 g. of chromium acetylacetonate, which can be recrystallized from chloroform-benzene.

Π.

 $\begin{array}{rcrc} Cr(NO_3)_3 \ + \ 3 \ C_5H_8O_2 \ = \ [(C_5H_7O_2)_3Cr] \ + \ 3 \ HNO_3 \\ (9 \ H_2O) \\ 400.2 \qquad 300.3 \qquad 349.3 \end{array}$ 

An alcoholic solution of  $[Cr(OH_2)_6](NO_3)_3$  is treated with the stoichiometric quantity of acetylacetone and then gently refluxed. The chromium complex crystallizes out after the excess alcohol is distilled off.

REFERENCES:

Red-violet crystals, m.p. 216°; can be sublimed in vacuum. Soluble in alcohol, chloroform and benzene; virtually insoluble in water and petroleum ether.

REFERENCES:

F. Hein. J. prakt. Chem. 153, 169 (1939); F. Gach. Monatsh. Chem. <u>21</u>, 108 (1900).—As far as preparation from chromium chloride hexahydrate and acetylacetone in the presence of urea is concerned, see W. C. Fernelius and F. E. Blanch in: T. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 130.

# Chromyl Chloride

# $CrO_2Cl_2$

I.  $K_2CrO_4 + 2 NaCl + 2 H_2SO_4 = CrO_2Cl_2 + Na_2SO_4 + K_2SO_4 + 2 H_2O_{194,2}$ 194.2 116.9 196.2 154.9

A clay crucible is used to fuse 200 g. of  $K_2CrO_4$  with 122 g. of NaCl at a temperature which should not be excessive. The melt is poured onto a sheet of iron and broken up into coarse pieces. These are placed in a 2-liter ground-joint flask and covered with 200 ml. of 100% H<sub>2</sub>SO<sub>4</sub>. A distilling condenser is connected to the flask at once, and a ground-joint receiving flask with a gas outlet tube is attached to the lower end of that condenser. When the initially vigorous reaction becomes moderate, the reaction flask is

heated gently until no further  $CrO_2Cl_2$  distills. The crude product is purified by a second distillation in a dry ground-glass apparatus; the pure  $CrO_2Cl_2$  is collected in dry glass ampoules, which are then melt-sealed.

II. 
$$K_2Cr_2O_7 + 4 \operatorname{NaCl} + 3 H_2SO_4 = 2 \operatorname{Cr}O_2Cl_2 + K_2SO_4 + 2 \operatorname{Na}_2SO_4 + 3 H_2O_294.2$$
  
294.2 116.9 294.2 309.8

It is possible to omit the fusion step. Thus, 150 g. of fuming  $H_2SO_4$  is added in portions to a mixture of 50 g. of NaCl and 80 g. of  $K_2Cr_2O_7$  (both thoroughly dried). Further procedure is the same as in method I. The yield is approximately 50%, based on  $K_2Cr_2O_7$ .

III. 
$$CrO_3 + 2 HCl = CrO_2Cl_2 + H_2O$$
  
100.0 72.9 154.9

A solution of 50 g. of  $CrO_3$  in 170 ml. of conc. HCl is prepared, and 100 ml. of conc.  $H_2SO_4$  is added in 20-ml. portions while cooling the flask in ice. The fluid mixture is poured into a separatory funnel, and after 20 minutes the lower  $CrO_2Cl_2$  layer is drained into a small ground-joint flask. Dry air is bubbled through it for several minutes and the crude  $CrO_2Cl_2$  is distilled as in method I.

PROPERTIES:

Deep-red liquid; fumes copiously in moist air. Should be stored in the dark and in sealed glass containers. M.p. -96.5 °C, b.p. 117 °C;  $d_{4}^{25}$  1.9118. Can react explosively with combustible organic and inorganic substances. Soluble in other inorganic acid chlorides and organic liquids, such as POCl<sub>3</sub>, CCl<sub>4</sub>, CHCl<sub>3</sub> and C<sub>6</sub>H<sub>6</sub>.

REFERENCES:

I and III. L. Vanino. Handb. d. Präp. Chemie [Handbook of Preparative Chemistry], I, Stuttgart, 1925; p. 713.

II. E. Moles and L. Gomez. Z. phys. Chem. 80, 513 (1912).

See also H. H. Sisler. Inorg. Syntheses, Vol. II, New York-London, 1946, p. 205.

# Chromium Trioxide-Pyridine

# CrO<sub>3</sub> · 2 py

#### $2 \mathbf{C}_5 \mathbf{H}_5 \mathbf{N} + \mathbf{CrO}_3 = 2 \mathbf{C}_5 \mathbf{H}_5 \mathbf{N} \cdot \mathbf{CrO}_3$

Four grams of  $CrO_3$  (0.04 mole) is dried in vacuum for four hours at 110°C and then chilled in ice-salt mixture. Fifty ml. of

pyridine (0.63 mole) is similarly chilled in a 300-ml. Erlenmeyer flask. The pyridine flask is agitated vigorously while situated in a cold bath, and the  $CrO_3$  is slowly added. The flask is then stoppered and shaken further until solution is complete (solution is hastened by the use of a large excess of pyridine). The cooling is necessary to prevent oxidation of the pyridine. The excess solvent is then removed in vacuum. The product is sensitive to light. Slow evaporation favors the formation of large crystals. Yield: 10.3 g. (100%).

PROPERTIES:

Yellow to dark-red crystals. Soluble in pyridine; insoluble in  $CCl_4$ , benzene and ether. Hygroscopic. Decomposes slowly at 100°C; at higher temperatures, burns to give voluminous green chromium oxide. Hydrolyzes at once with water. Stable indefinitely in the dark. Stored in sealed containers at room temperature.

REFERENCES:

H. H. Sisler, J. D. Bush and O. E. Accountius. J. Amer. Chem. Soc. <u>70</u>, 3827 (1948); O. E. Accountius, J. D. Bush and H. H. Sisler in: J. C. Bailar, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 94.

# Chromyl Nitrate

# $CrO_2(NO_3)_2$

$$CrO_3 + N_2O_5 = CrO_2(NO_3)_2$$
  
100.0 108.0 208.0

A powder funnel is used to rapidly pour 8.3 g. of  $N_2O_5$  into a 50-ml. ground-joint flask precharged with 7 g. of  $CrO_3$  and a few (vacuum) boiling stones. The flask is attached to a distillation apparatus whose joints are lubricated with silicone grease and which is protected against entry of atmospheric moisture by means of a  $P_2O_5$  tube. The reaction begins after a short time, with fusion of the solids. The reaction mixture should be left standing overnight at room temperature.

The dark-red liquid product is distilled in aspirator vacuum. A liquid-nitrogen-cooled trap is interposed between the apparatus and the aspirator to prevent access of moisture and to condense the NO<sub>2</sub> and N<sub>2</sub>O<sub>5</sub> which distill off. The  $CrO_2(NO_3)_2$  distills at a bath temperature of about 75°C (partial decomposition). The receiver then contains 5.8 g. of pure  $CrO_2(NO_3)_2$ .

PROPERTIES:

Dark-red liquid, sensitive to moisture. M.p. -27°C, b.p. (10<sup>-3</sup> mm.) 28°C; (17 mm.) 67°C. Decomposes at about 120°C.

REFERENCES:

M. Schmeisser and D. Lützow. Angew. Chem. <u>66</u>, 230 (1954); D. Lützow. Thesis, Univ. München, 1955.

#### **Chromyl Perchlorate**

#### $CrO_2(ClO_4)_2$

 $\begin{array}{rrrr} CrO_3 &+& 2\,Cl_2O_3 &=& CrO_2(ClO_4)_2 &+& 2\,ClO_2 &+& 1/_2\,O_2 \\ 100.0 && 333.8 && 282.9 && 134.9 \end{array}$ 

A two-neck flask is used; then, at  $-50^{\circ}$ C, 5 g. of Cl<sub>2</sub>O<sub>6</sub>, followed by 3 g. of CrO<sub>3</sub>, is added through one neck. This neck is then closed off either with a ground stopper lubricated with fluorinated hydrocarbon grease (see under Cl<sub>2</sub>O, p. 299 f.), or by sealing off. The other neck leads to a manifold carrying sealable ampoules and a second, similar flask. The open end of the manifold is closed off with a P<sub>2</sub>O<sub>5</sub> tube.

The cold bath is now replaced with a bath at  $+6^{\circ}$ C. The Cl<sub>2</sub>O<sub>6</sub> melts, and the two components react vigorously. The reactor is allowed to stand at 0°C for several hours (preferably overnight). After this, no further gases are evolved.

The reactor is now cooled with liquid nitrogen and the entire system evacuated to about 0.1 mm. The cold bath is removed and the second flask (at the manifold) is cooled; within a few minutes,  $Cl_2$  and  $ClO_2$  distill with foaming. To remove these gases completely, the vessel is immersed in a bath at +20°C for one half hour and vacuum is applied. As soon as no further volatiles distill, the bath temperature is raised to about 35-36°C. The  $CrO_2(ClO_4)_2$ now distills into the manifold and flows into the first ampoule (transparent red liquid). The manifold with the ampoules should be somewhat inclined. When sufficient compound has collected in the first ampoule, the latter is sealed off. Additional distilled product collects in the stub left from the first ampoule, and is driven into the next ampoule by heating with a hot-air blower.

PROPERTIES:

Red liquid, very sensitive to moisture. M.p.  $-1^{\circ}$ C, b.p. (extrapolated) (760 mm.) 174.7°C; (0.08 mm.) 35°C; (0.8 mm.) 45°C. Powerful oxidant; dissolves in CCl<sub>4</sub>. May be stored for months in the dark at Dry Ice temperature. Often explodes at +80°C.

REFERENCES:

M. Schmeisser. Angew. Chem. <u>67</u>, 493 (1955); D. Lützow. Thesis, Univ. München, 1955.

### **Rubidium Chromate**

#### Rb<sub>2</sub>CrO<sub>4</sub>

I.

 $Rb_2CO_3 + CrO_3 = Rb_2CrO_4 + CO_2$ 231.0 100.0 287.0

Obtained by evaporation of an aqueous solution of  $CrO_3$  which has been neutralized with  $Rb_2CO_3$  (or RbOH).

The by-product  $Rb_2Cr_2O_7$  forms at even a very small excess of  $CrO_3$ ; therefore somewhat more than the stoichiometric quantity of  $Rb_2CO_3$  should be used.

II. Preparation analogous to that of Cs<sub>2</sub>CrO<sub>4</sub>.

PROPERTIES:

Yellow, rhombic crystals, isomorphous with  $K_2CrO_4$  and  $K_2SO_4$ . Readily soluble in water (42% at 20°C).

REFERENCES:

L. Grandeau. Ann. Chim. Phys. (3) <u>67</u>, 228 (1863); J. W. Retgers. Z. phys. Chem. <u>8</u>, 39 (1891); Abeggs Handbuch der anorg. Chemie [Abegg's Handbook of Inorganic Chemistry], IV, 1, p. 362 (1921).

#### **Rubidium Dichromate**

#### Rb<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub>

I.

 $Rb_{2}CO_{3} + 2 CrO_{3} = Rb_{2}Cr_{2}O_{7} + CO_{2}$ 231.0 200.0 387.0

Obtained by evaporation of stoichiometric mixtures of  $Rb_2CO_3$  (or RbOH) and  $CrO_3$ .

II. Preparation analogous to that of  $Cs_2Cr_2O_7$ .

PROPERTIES:

Trimorphic; forms A and B deposit together from solution above  $35^{\circ}$ C. Orange-colored monoclinic or red triclinic crystals. Moderately soluble in water (5% at  $18^{\circ}$ C).

**REFERENCES:** 

L. Grandeau. Ann. Chim. Phys. (3), <u>67</u>, 227 (1863); Abeggs Handbuch der anorg. Chemie IV, 1, p. 362 (1921).

#### **Cesium Chromate**

#### Cs<sub>2</sub>CrO<sub>4</sub>

A small excess of  $Ba(OH)_2$  is added to a warm solution of  $Cs_2Cr_2O_7$ . The sparingly soluble  $BaCrO_4$  is filtered off and the solution is concentrated until crystallization occurs.

PROPERTIES:

Yellow hexagonal or rhombic crystals, readily soluble in water.

**REFERENCE:** 

J. H. de Boer, J. Broos and H. Emmens. Z. anorg. allg. Chem. <u>191</u>, 113 (1930).

#### Cesium Dichromate

 $Cs_2Cr_2O_7$ 

 $(NH_4)_2Cr_2O_7 + 2CsCl = Cs_2Cr_2O_7 + 2NH_4Cl$ 252.1 336.7 481.8

Reaction of warm solutions of  $(NH_4)_2Cr_2O_7$  and CsCl, followed by cooling, yields orange-red crystals of  $Cs_2Cr_2O_7$ , which are still contaminated with about 5% of  $(NH_4)_2Cr_2O_7$ . To decompose the ammonium salt, the product is calcined at a low temperature. Recrystallization gives an excellent yield of pure  $Cs_2Cr_2O_7$ .

PROPERTIES:

Orange-red triclinic crystals; sparingly soluble in cold, readily soluble in hot water.

REFERENCE:

J. H. de Boer, J. Broos and H. Emmens. Z. anorg. allg. Chem. <u>191</u>, 113 (1930).

### Potassium Fluorochromate

### K[CrO<sub>3</sub>F]

 $\begin{array}{rrrr} K_2 Cr_2 O_7 &+ & 2 \ HF &= & 2 \ K[CrO_3 F] &+ & H_2 O \\ 294.2 & & 40.0 & & 316.2 \end{array}$ 

Powdered  $K_2Cr_2O_7$  is heated in a Pt dish with excess of conc. HF until solution is complete. On cooling,  $K[CrO_3F]$  separates as red crystals.

PROPERTIES:

Formula weight 158.11. Ruby-red bipyramids, readily soluble in water. Etches glass vessels in which it is stored. Crystal structure: tetragonal (space group  $C_{4h}^{\delta}$ ).

**REFERENCES:** 

 A. Streng. Liebigs Ann. <u>129</u>, 227 (1864); J. A. A. Ketelaar and E. Wegerif. Recueil Trav. Chim. Pays-Bas <u>57</u>, 1269 (1938).

### Potassium Chlorochromate

#### K[CrO<sub>3</sub>Cl]

 $\begin{array}{rrrr} K_{2}Cr_{2}O_{7} &+ & 2 \ HCl &= & 2 \ K[CrO_{3}Cl] &+ \ H_{2}O \\ & & 294.2 & & 72.9 & & 349.1 \end{array}$ 

I. Fifty grams of fine  $K_2Cr_2O_7$  powder is dissolved in a mixture of 65 ml. of conc. HCl and 50 ml. of water (by heating to 70°C). The solution is filtered through a jacketed funnel heated with hot water. After 1-2 days, the nascent crystals are filtered off with suction, recrystallized from glacial acetic acid, and dried in a vacuum desiccator over  $H_2SO_4$ .

II. 
$$K_2CrO_4 + CrO_2Cl_2 = 2 K[CrO_3Cl]$$
  
194.2 154.9 349.1

A three-neck flask is fitted with a stirrer, a thermometer, a dropping funnel, and a gas outlet tube. A solution of 75 g. of

 $K_2CrO_4$  in 125 ml. of hot water is placed in the flask, and 86 g. of  $CrO_3Cl_2$  is added dropwise with stirring. The temperature is held at 90-100 °C by means of a Bunsen burner. Stirring is continued for 1 hour at the same temperature and the flask contents are then poured into a beaker. After 18 hours the nascent crystals are filtered off with suction and pressed together firmly to remove the mother liquor as thoroughly as possible without washing. The product is then placed on a clay plate, covered with a watch glass, and allowed to stand for 10 hours. Yield: 109 g. (81%, based on  $K_2CrO_4$ ).

The mother liquor, cooled at 0°C for 1.5 hours, yields about 16 g. of less pure K[CrO<sub>3</sub>Cl]. To purify this, 30 g. of the impure product is dissolved in 100 ml. of acetone, filtered, and 700-800 ml. of CCl<sub>4</sub> is added slowly with stirring; 16-17 g. of pure K[CrO<sub>3</sub>Cl] is thus obtained.

#### PROPERTIES:

Formula weight 174.56; d 2.497. Sparkling, orange-colored crystalline needles, soluble in glacial acetic acid and acetone. Undergoes hydrolytic cleavage in water. Heating the salt to 100°C causes loss of chlorine.

**REFERENCES**:

- I. L. Vanino. Handb. d. Präp. Chemie [Handbook of Preparative Chemistry], I, Stuttgart, 925, p. 321.
- II. H. H. Sisler in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 208.

#### Potassium Tetraperoxochromate (V)

#### K<sub>3</sub>CrO<sub>8</sub>

Prepared from KOH,  $CrO_3$  and  $H_2O_2$ .

A mixture of 25 ml. of 50% CrO<sub>3</sub> solution, 100 ml. of 25% KOH, and 100 ml. of  $H_2O$  is cooled in a cold bath until ice begins to form. Now, 30 ml. of 30%  $H_2O_2$  is added dropwise (with shaking), care being taken to keep the solution temperature from rising above 0°C. The initially red-yellow solution soon acquires a black-brown color. The salt which drops to the bottom of the vessel after 1-2 hours is filtered off with suction, washed with 95% alcohol until the washings are colorless, then with ether, and stored in a stoppered vessel. The yield is about 50%, based on the  $H_2O_2$  used. PROPERTIES:

Formula weight 297.30. Red-brown crystals, which may be stored for months without decomposition. Moderately soluble in cold water, insoluble in alcohol and ether.

**REFERENCE**:

E. H. Riesenfeld, H. E. Wohlers and W. A. Kutsch. Ber. dtsch. chem. Ges. <u>38</u>, 1887 (1905).

#### Ammonium Pentaperoxodichromate

 $(NH_4)_2 Cr_2 O_{12} \cdot 2 H_2 O$ 

Prepared from NH<sub>4</sub>Cl, CrO<sub>3</sub> and H<sub>2</sub>O<sub>2</sub>.

The procedure for the blue ammonium salt is the same as that used for  $K_3CrO_8$  (see above). The quantities used are: 100 ml. of  $H_2O$ , 5 ml. of conc. HCl, 10 g. of NH<sub>4</sub>Cl, 10 ml. of 50% CrO<sub>3</sub> solution, and 25 ml. of 30%  $H_2O_2$ . At the end, the product is washed only briefly with 90% alcohol.

**PROPERTIES:** 

Formula weight 386.13. Violet-black crystalline powder consisting of flat prisms which show strong pleochroism (bright redbrown and dark blue-violet). May be stored for a few days in a cold desiccator; transforms completely to  $(NH_4)_2CrO_4$  on 24-hour exposure in the air; decomposes explosively at 50°C to  $Cr_2O_3$ . Soluble in ice water (violet-brown color).

**REFERENCES**:

E. H. Riesenfeld, H. E. Wohlers and W. A. Kutsch. Ber. dtsch. chem. Ges. <u>38</u>, 1888 (1905); O. F. Wiede. Ber. dtsch. chem. Ges. <u>31</u>, 518 (1898); R. Schwarz and H. Giese. Ber. dtsch. chem. Ges. <u>66</u>, 310 (1933).

#### Diperoxotriamminechromium (IV)

### (NH<sub>3</sub>)<sub>3</sub>CrO<sub>4</sub>

Prepared from ammonia, CrO<sub>3</sub> and H<sub>2</sub>O<sub>2</sub>.

A mixture of 25 ml. of 10% ammonia and 5 ml. of 50%  $CrO_3$  solution is treated dropwise at 0°C with 5 ml. of 30%  $H_2O_2$ . The

resultant solution is first allowed to stand for one hour in a cooling mixture, and is then heated (together with the copious precipitate of  $(NH_4)_2Cr_2O_7$  contained therein) to about 50 °C until the vigorous evolution of gas ceases and the salt dissolves almost completely. Finally, the solution is filtered and cooled once more to 0 °C. The  $(NH_3)_3CrO_4$  which crystallizes is filtered off with suction, washed with absolute alcohol and ether, and dried in a desiccator over KOH. Yield: about 0.3 g.

SYNONYMS:

Chromium tetroxide triammine or triamminechromium tetroxide.

PROPERTIES:

Formula weight 167.11; d<sup>15.8</sup> 1.964. Light-brown needles, soluble in dilute ammonia and water (partial decomposition). Insoluble in other solvents. The product should be protected from moisture, but because of the danger of explosion, storage ampoules other than the type sealed by fusion of the outlet should be used.

REFERENCES:

E. H. Riesenfeld. Ber. dtsch. chem. Ges. <u>38</u>, 4070 (1905); O. F. Wiede. Ber. dtsch. chem. Ges. <u>30</u>, 2180 (1897); for discussion of valence state, see S. S. Bhatnagar, B. Prakash and A. Hamid. J. Chem. Soc. (London) <u>1938</u>, 1432.

#### Barium Orthochromate (IV)

#### Ba<sub>2</sub>CrO<sub>4</sub>

Stoichiometric quantities of the starting materials, which must be very pure and anhydrous, are thoroughly mixed. (However, a very small excess of  $Ba(OH)_2$ , i.e., 0.03-0.06 moles BaO/atom Cr, must be provided. Any larger excess of the base will give rise, in the heating which follows, to partial or sometimes complete formation of tribarium chromate (IV),  $Ba_3CrO_5$ . The latter is a heavy, blackish-green, glittering crystalline powder which appears olive-brown under the microscope.)

About 4 g. of the mixture is then heated in a sintered alumina boat in an  $O_2$ -free nitrogen stream (2 hours at 900-950°C).
PROPERTIES:

Microcrystalline, heavy emerald green powder. Readily soluble in dilute HCl or HClO<sub>4</sub>, even in the cold, with brownish yellow color. Water causes hydrolysis. Stable to methanol.

**REFERENCE:** 

R. Scholder and G. Sperka. Z. anorg. allg. Chem. 285, 49 (1956).

#### Barium Chromate (V)

Ba<sub>3</sub>(CrO<sub>4</sub>)<sub>2</sub>

An intimate mixture of 1 mole of  $BaCrO_4$  and 0.50 moles of  $BaCO_3$  is heated in an  $O_2$ -free nitrogen stream at 1000 °C. Four hours of heating suffices for about 2 g. of reactants. The  $Ba_3(CrO_4)_2$  product is of excellent purity.

PROPERTIES:

Black-green microcrystalline powder. Water causes gradual decomposition. Completely soluble in dilute acids, with disproportionation to Cr (III) and Cr (VI).

**REFERENCE:** 

R. Scholder and W. Klemm. Angew. Chem. 66, 463 (1954).

#### Sodium Thiochromite

NaCrS<sub>2</sub>

Prepared by reaction of  $K_2CrO_4$  with a soda-sulfur melt.

An intimate mixture of 1 part of  $K_{2}CrO_{4}$  with 30 parts of KNaCO<sub>3</sub> and 30 parts of sulfur is heated for 30-60 minutes in a covered sintered alumina crucible; the latter is placed in an electric furnace. The temperature is 750-850 °C. After heating, the crucible is allowed to cool slowly. The cold melt is slurried in water, then washed by decantation several times with dilute NaOH. The thiochromite is filtered off and thoroughly washed, first with dilute alcoholic NaOH, then with pure alcohol, and finally with ether. The product is free of potassium despite the use of K salts.

PROPERTIES:

Formula weight 139.13; d 3.2. Crystalline gray-black aggregate with a greenish luster. Well-formed hexagonal leaflets are produced above 800°C; these appear garnet-red by transmitted light. When moist, rapidly darkens and decomposes on exposure to air.

REFERENCES:

W. Rüdorff and K. Stegemann. Z. anorg. allg. Chem. <u>251</u>, 379 (1943); R. Schneider. J. prakt. Chem. <u>56</u>, 415 (1897).

#### Dibenzenechromium (0)

#### (C<sub>6</sub>H<sub>6</sub>)<sub>2</sub>Cr

 $\begin{array}{rrrr} & \operatorname{AlCl}_{3} & + & 2\operatorname{Al} & + & \operatorname{AlCl}_{3} & + & 6\operatorname{C_6H_6} \rightarrow 3 \left[\operatorname{Cr}(\operatorname{C_6H_6})_2\right] \left[\operatorname{AlCl}_{4}\right] \\ & & 475.1 & 53.9 & 133.3 & 468.6 \end{array}$   $2 \left[\operatorname{Cr}(\operatorname{C_6H_6})_2\right]^+ + & \operatorname{S_2O_4^{2-}} + & 4\operatorname{OH}^- \rightarrow & 2\operatorname{Cr}(\operatorname{C_6H_6})_2 + & 2\operatorname{SO_3^{2-}} + & 2\operatorname{H_2O} \\ & & 416.5 \end{array}$ 

A 250-ml. three-neck flask is used, and 25 g. (0.16 moles) of anhydrous  $CrCl_3$ , 3.5 g. (0.13 moles) of dry Al powder, and 60 g. (0.45 moles) of sublimed and rapidly ground  $AlCl_3$  are weighed in. The flask is evacuated several times with an aspirator and refilled with dry,  $O_2$ -free nitrogen. Then, 150 ml. of absolute benzene is introduced in a countercurrent stream of inert gas. The flask is again evacuated for several minutes to remove traces of HCl (with the evaporating benzene). Then, 10 drops (0.3 ml.) of mesitylene are added under protection of the N<sub>2</sub> blanket. The flask is now fitted with a reflux condenser carrying a mercury pressurerelief valve. A high-speed Hg-seal stirrer and a stopper are placed on the other necks.

The N<sub>2</sub> stream is cut off and the mixture is refluxed for 35-40 hours with vigorous stirring. The flask contents are cooled, then decomposed by pouring slowly (under N<sub>2</sub>) into 200 ml. of CH<sub>3</sub>OH contained in a 4-liter three-neck flask. The latter is cooled in ice; two of the necks are fitted with stopcocks and the third carries a high-speed Hg-seal stirrer. Then 200 ml. of H<sub>2</sub>O is also added. When hydrolysis is complete, 2 liters of benzene is added, then a solution of 220 g. of KOH in 500 ml. of H<sub>2</sub>O. Finally, 220 g. of solid sodium dithionite is rapidly introduced. After 2 hours of high-speed stirring, the dark-brown solution is carefully filtered with suction (in the absence of air) into an evacuated 3-liter flask, which carries an  $N_2$  inlet tube. The solution is dried with solid KOH. It is then transferred (under  $N_2$ ) to a distillation apparatus, and the solvent is removed thoroughly on a hot water bath. The solid black residue is washed 3 times with absolute ether (under  $N_2$ ) and then sublimed in high vacuum at 160°C. Yield: 29.5 g. (90% of theoretical).

PROPERTIES:

Black, diamagnetic crystals, sensitive to air. M.p. 284-285°C. Slightly soluble in ether and petroleum ether, giving a brown color; moderately soluble in benzene.

**REFERENCES**:

E. O. Fischer and W. Hafner. Z. Naturforsch. <u>10b</u>, 665 (1955); Z. anorg. allg. Chem. <u>286</u>, 146 (1956); E. O. Fischer, W. Hafner and J. Seeholzer. Private communication.

# Bis(diphenyl)chromium (0)

 $(C_{12}H_{10})_2Cr$ 

# $[(C_{12}H_{10})_2Cr]^+ \xrightarrow{S_2O_*^{2-}} (C_{12}H_{10})_2Cr$

The melt is prepared and hydrolyzed in the same manner as described below for bis(diphenyl)chromium (I) iodide. The first filtrate is rejected.

An excess of alkaline sodium dithionite solution is added under  $N_2$  to the later, pure orange-red filtrates, whereupon the bis(diphenyl)chromium (I) cation is reduced instantly and precipitates as bis(diphenyl)chromium (0). After standing for one half hour the precipitate is filtered off on a large, fine fritted-glass funnel and then dried for 1-2 days over  $P_2O_5$ . It is then extracted with ether or pentane in the absence of air; the bis(diphenyl)chromium separates from the solvent in beautiful small crystals. These are filtered off, dried in vacuum, and stored under  $N_2$ . Yield: 10 to 12 g.

PROPERTIES:

Crystals with brasslike luster. M.p. (not sharp) at 112°C. Soluble in ether, alcohol, benzene, etc.; diamagnetic. Limited stability in air when dry. Dibenzenechromium (0) and diphenylbenzenechromium (0) can be prepared from the corresponding chromium (I) salts in esentially the same way.

REFERENCES:

E. O. Fischer and D. Seus. Chem. Ber. <u>89</u>, 1814 (1956); F. Hein and W. Kleinwächter. Private unpublished communication.

#### Dibenzenechromium (I) lodide

#### $[(C_6H_6)_2Cr]I$

Ten grams (0.05 moles) of  $Cr(C_{6}H_{6})_{2}$  fine powder is shaken with 200 ml. of benzene and 100 ml. of  $H_{2}O$  in a separatory funnel, while air is passed through, until all the solid dissolves and the benzene phase becomes virtually colorless. The yellow-brown aqueous layer is filtered and treated with saturated aqueous KI solution (stirring) until no further yellow precipitate separates out. After cooling in ice, the precipitate is filtered off, washed 3 times with some  $C_{2}H_{5}OH$ , and finally with ether. It is then dried in vacuum. Yield: 10.5 g., or 65% based on  $Cr(C_{6}H_{6})_{2}$ .

PROPERTIES:

Egg-yellow, stable in air, moderately soluble in  $H_2O$ .

REFERENCES:

E. O. Fischer and W. Hafner. Z. anorg. allg. Chem. <u>286</u>, 146 (1956); E. O. Fischer. Private communication.

## Bis(diphenyl)chromium (I) Iodide

## $[Cr(C_{12}H_{10})_2]I$

Ten grams of sieved, anhydrous  $CrCl_3$  dust, 8 g. of Al powder, 27 g. of sublimed diphenyl, and 30 g. of AlCl<sub>3</sub> (powdered in a mortar) are separately dried for one half hour in an oven at 110°C. Then, the  $CrCl_3$  is mixed intimately with the Al powder, and the biphenyl with the AlCl<sub>3</sub>. The two mixtures are then blended thoroughly with each other in a 150-ml. beaker placed in a drying oven. Finally, the total mixture is covered with a layer of pure biphenyl (3-4 g.), and the beaker is covered with a watch glass. The beaker is now placed in a silicone oil bath preheated to 100°C, and the bath temperature is slowly raised to 110°C. As soon as the reaction begins (110-120°C, melting of the mass, followed by puffing up and evolution of HCl vapors), the heating of the oil bath is discontinued. The heat of reaction causes the temperature of the mixture to rise spontaneously to 140-150 °C. After 10 minutes the melt is stirred vigorously with a thermometer, care being taken to keep the temperature from rising above 160 °C. The reaction is allowed to complete itself in one half hour. During this time, the oil bath temperature is kept at 120 °C. Careful conduct of the melt reaction is most important in this preparation.

The beaker with the melt is now cooled to room temperature. The melt is added with a spatula to 100 ml. of methanol (addition in portions) in an 800-ml. beaker placed in an ice bath. An orange-red to brown solution forms; cold, saturated NaCl solution is then added with stirring. This yields an easily filtered product which is separated once on a 15-cm. Buchner funnel. Then about 3 g. of solid KI is added to the acidic, darkbrown filtrate: this causes precipitation of the bis(diphenyl)chromium (I) cation present. However, most of the product is in the filtration residue, and is obtained by leaching the residue several times (on the funnel) with 100-ml. portions of water, followed by suction-drying. Before each leaching, a fast stream of air is drawn through the filter cake for 10-15 minutes, in order to oxidize any remaining chromium (0) to the monovalent state. The leaching is discontinued when the wash water becomes almost colorless. The bis(diphenyl)chromium (I) iodide is reprecipitated by stirring about 8-10 g. of solid KI into the filtrate. It is filtered off with suction and washed with water, then 10 ml. of alcohol and two 10-ml. portions of ether. Yield: 22-25 g. The crude product is already very pure; it can be recrystallized from alcohol.

## PROPERTIES:

Formula weight 487.33; m.p. 157°C. Depending on size, orange to reddish black crystals. Soluble in pyridine, alcohol, chloroform, acetone; almost insoluble in benzene and water; insoluble in ether and naphtha. Not sensitive to dilute hydrochloric acid.

#### **REFERENCES:**

F. Hein. Ber. dtsch. chem. Ges. <u>54</u>, 2716 (1921); E. O. Fischer and D. Seus. Chem. Ber. <u>89</u>, 1814 (1956); F. Hein and W. Kleinwächter. Unpublished private communication.

# (Diphenyl)(benzene)chromium (I) Iodide

# $[(C_{12}H_{10})Cr(C_6H_6)]I$

All operations are conducted under pure  $N_{\mbox{\scriptsize 2}}$  in the absence of moisture.

A Grignard solution is prepared from 37 g. of magnesium, 235 g. of bromobenzene and 900 ml. of absolute ether. After the

end of the reaction, the solution is decanted from the unreacted Mg into a 1.5-liter sulfonation flask, which is provided with a stirrer, a thermometer, a tap-injection bulb from which solid reagents can be added, and inlet and outlet tubes for  $N_{2^*}$ 

The solution is cooled to -15 to -18 °C. Vigorous stirring and good cooling are provided, and 40 g. of sublimed CrCl<sub>3</sub> is tapped from the bulb into the flask at a uniform rate; total addition time: 2 to 3 hours. (The CrCl<sub>3</sub> is preextracted with boiling HCl, washed, dried, and sieved through a U.S. standard 60-mesh screen.) The reaction temperature should not rise above  $-12^{\circ}$ C. The mixture becomes black-brown. After the addition of CrCl<sub>3</sub>, stirring is continued 2 to 3 hours. After standing overnight in an ice chest, the mixture is stirred thoroughly and decomposed by pouring it slowly onto an ice-H<sub>2</sub>SO<sub>4</sub> mixture (750 g. of ice, 25 ml. of conc.  $H_2SO_4$ ) contained in a 4-liter breaker. The addition proceeds in air and with constant stirring while the beaker is immersed in an ice-salt bath. The yellow-red ethereal emulsion is rapidly decanted into a dish and the ether is driven off. The aqueous layer is filtered through a suction funnel with the largest possible filtering surface.

The residue from the ethereal layer is stirred with approximately 50 ml. of 50% KI and 50 ml. of saturated Na<sub>2</sub>SO<sub>3</sub> solution, and is then thoroughly extracted with chloroform (shaking in a separatory funnel) until the solvent is only slightly yellow. The aqueous solution, which contains KI, is combined with the filtrate from the aqueous layer and similarly extracted with chloroform. The residue from the filtration of the aqueous layer, the filter paper, the funnel, and all vessels are also extracted with chloroform. The combined orange-colored chloroform extracts (which contain the crude iodide) are washed twice, each time with 15 ml. of KI and 10 ml. of Na<sub>2</sub>SO<sub>3</sub> solutions, then once with 25 ml. of H<sub>2</sub>O, and dried for 24 hours over anhydrous potassium carbonate. The filtered chloroform solution is concentrated, under anhydrous conditions, in aspirator vacuum at a bath temperature of 25 to 30°C. The residual viscous mass is rinsed into a dish with a minimum amount of chloroform. To remove diphenyl, the material is triturated, first with 100-ml. portions and later with 30-ml. portions of absolute ether, until a sample of extract shows almost no residue on evaporation. A total of 1.5 to 2 liters of absolute ether is required. The viscous, red-orange, crude iodide hardens and becomes powdery as the extraction of diphenyl progresses. It is dried over  $P_2O_5$  in vacuum. It may be kept for months if stored in a cool place away from light. Yield: 35-40 g. of crude iodide. It is composed of bis(diphenyl)chromium (I) iodide, (diphenyl)(benzene)chromium (I) iodide (the principal constituent), and a very small percentage of dibenzenechromium (I) iodide.

To obtain pure (diphenyl) (benzene) chromium (I) iodide, the procedure is as follows: 39 g. of crude iodide is dissolved in 900 ml. of methanol, treated with 300 ml. of water, and passed through an anion exchange column (e.g., OH form of Wofatit L 150 or Amberlite IRA 410; 150 g. of dry material in 70% methanol) at a rate of 3 ml./min. The column is then eluted with 70% alcohol and the vellow to orange fraction of the filtrate is collected in the absence of CO<sub>2</sub>. This fraction is concentrated in aspirator vacuum at a bath temperature of 30-35°C until a methanol-free solution remains. This is filtered to remove a slight cloudiness (diphenyl). The clear, filtered solution is diluted to 300 ml. with water and treated with a solution of 18 g. of anthranilic acid (m.p. 145°C) and 12 g. of KOH in 60 ml. of H<sub>2</sub>O while cooling the flask in ice; this treatment causes the (C12H10)2Cr (I) anthranilate to deposit as an orange-yellow, amorphous precipitate. After standing for 3 hours, the precipitate is removed by filtration through a very fine fritted-glass funnel. The filtrate is treated with 20 g. of solid KI in a separatory funnel. The  $(C_{12}H_{10})$ -(C<sub>e</sub>H<sub>e</sub>)CrI separates at once as an oil. The oil is extracted with chloroform until the latter is only slightly yellow. The combined chloroform extracts are thoroughly shaken with some 20% KI solution, and then with a very small quantity of water; the extracts are then dried for several hours over potassium carbonate. The filtered solutions are concentrated by distilling off the chloroform in vacuum (under anhydrous conditions) at a bath temperature of 30-35°C. Finally, about a 10-fold quantity of absolute ether is added, causing an orange-red oil to separate; this gradually solidifies and can be ground under ether. The supernatant ether layer is replaced 2 or 3 times to remove the chloroform. The powder is filtered off under anhydrous conditions, washed several times with ether, and dried in a drying pistol at 2 mm. and 55-60°C (using acetone as the heating medium). The product is recrystallized by dissolving in absolute alcohol at 60-70°C (anhydrous conditions), filtering through a very fine fritted-glass funnel while still hot, and storing overnight at -20°C. This yields massive orange-red crystals. The (C12H10)(C6H6)CrI can also crystallize in goldenyellow hexagonal leaflets, but these transform into the orange-red crystals after standing for several days in the mother liquor.

Since the  $(C_{12}H_{10})(C_{6}H_{6})$ CrI often separates as an oil, seeding the solution may be helpful. The precipitation can be completed by very slow addition of a 3- to 5-fold quantity of ether. The precipitate is filtered off under anhydrous conditions, washed twice with some 1:1 absolute ether/absolute ethanol, twice with absolute ether, and then dried to a constant weight in a drying pistol at 2 mm. and 55-60°C (using acetone as the heating medium). This drying quantitatively removes the ether, which otherwise adheres tenaciously. Yield: 26 g. of  $(C_{12}H_{10})(C_{6}H_{6})$ CrI. The (diphenyl)(benzene)chromium (I) iodide can also be prepared by a reductive Friedel-Crafts reaction, starting with an appropriate mixture of benzene, diphenyl,  $CrCl_3$ ,  $AlCl_3$  and Alpowder. The reaction can be carried out under reflux at atmospheric pressure, but again affords (dipheny)(benzene)chromium in a mixture with dibenzenechromium and bis(diphenyl)chromium. After conversion to the iodides, they must be separated from each other in a manner analogous to that given above.

The yield in this method, even under the most favorable conditions, is lower than that of the Grignard procedure.

PROPERTIES:

Formula weight 411.33. Red-orange, massive, somewhat lightsensitive crystals. M.p. approximately 160°C (decomp.). Can be stored in vacuum or under  $N_2$  in the dark. Readily soluble in pyridine; soluble in chloroform, alcohol and acetone; less soluble in water; insoluble in ether, benzene and petroleum ether.

REFERENCES:

F. Hein. Ber. dtsch. chem. Ges. <u>54</u>, 2741 (1921); F. Hein and H. Meininger. Z. anorg. allg. Chem. <u>145</u>, 115 (1925); F. Hein and E. Markert. Ber. dtsch. chem. Ges. <u>61</u>, 2261 (1928); F. Hein, P. Kleinert and E. Kurras. Z. anorg. allg. Chem. <u>289</u>, 229 (1957); H. H. Zeiss and M. Tsutsui. J. Amer. Chem. Soc. <u>79</u>, 3062 (1957); F. Hein and K. Eisfeld. Z. anorg. allg. Chem. <u>292</u>, 162 (1957).

## Molybdenum

Мо

I.

The  $MoO_3$  is obtained by heating ammonium molybdate. Since  $MoO_3$  is volatile at higher temperatures, it is prereduced in a stream of hydrogen at about 500 °C to the nonvolatile lower oxides. The oxides are then reduced to the metal at about 1000 °C. The product is allowed to cool in an H<sub>2</sub> stream, and the metal is obtained as a gray-black powder.

II. 
$$3 \operatorname{MoO}_{2} + 4 \operatorname{Al} = 3 \operatorname{Mo} + 2 \operatorname{Al}_{2} \operatorname{O}_{3}$$
  
383,9 107.9 287.9 203.9

Because of the volatility of  $MoO_3$ , the starting material is  $MoO_2$ , which is obtained by reduction of  $MoO_3$  with  $H_2$  at

dark-red heat. A clay crucible embedded in dry sand is used, and a mixture of 80 g. of  $MoO_2$  and 21 g. of Al powder (or, better, Al granules the size of grains of sand) is placed in it. The mixture is caused to react by means of an ignition mixture.\* After cooling the crucible is broken up, and the solid melt of MO, weighing about 50 g. (about 90% yield), is solated. If 60 g. or 40 g. of  $MoO_2$  is charged, the yield drops to 70-80%. The metal contains 98-98.5% Mo as well as some Si, Fe and Al.

PROPERTIES:

Solid Mo is bright, with a silvery luster. Powder is light- to black-gray, depending on particle size. M.p. 2620°C; d 10.23; hardness 5.5. Attacked (with difficulty) by nonoxidizing acids and aqueous alkalies. Crystal structure: A 2 type.

REFERENCES:

H. Funk. Darst. der Metalle im Laboratorium [Preparation of Metals in the Laboratory], Stuttgart, 1938, p. 69 f.; H. Biltz and R. Gärtner. Ber. dtsch. chem. Ges. <u>39</u>, 3370 (1906).

# Dibenzenemolybdenum (0)

## $(C_6H_6)_2M_0$

An intimate mixture of 4 g. (0.015 moles) of  $MoCl_5$ , 3 g. (0.023 moles) of anhydrous fine  $AlCl_3$  powder and 1 g. (0.04 g.-atoms) of Al powder is placed in a glass combustion tube of about 75-ml. capacity. About 30 ml. of absolute benzene is then added. The tube is evacuated, sealed, placed in an iron protective tube, and heated in a horizontal position for 15 hours at 120 °C.

After cooling, the tube is carefully opened and the dark-colored contents are decomposed with 20 ml. of methanol (cooling) and then treated with 75 ml. of water. The residue is filtered off on a

<sup>\*</sup>Ignition mixture (German "Zündgemisch" or "Zündkirsche"-ignition cherry) is made from 15 parts by weight of barium peroxide and 2 parts of powdered magnesium metal, intimately mixed and held together with collodion. The whole is wrapped with magnesium ribbon, a piece of the ribbon serving as the fuse. Magnesium burns with the evolution of much heat; the barium peroxide furnishes the large amounts of oxygen needed for such forced combustion (H. Blücher, Auskunftsbuch für die chemische Industrie [Data Book for the Chemical Industry], 18th ed., de Gruyter, Berlin, 1954, p. 1314).

fritted-glass funnel. The dark-colored filtrate is transferred to a 500-ml. three-neck flask, carefully prepurged with N<sub>2</sub>. The solution is covered with 200 ml. of benzene, and 10 g. of potassium diaminomethanedisulfinate  $(NH_2)_2C(SO_2K)_2$  [or the same amount of formamidinesulfinic acid  $(NH_2)_2CSO_2$ ] is added with vigorous stirring. Next, 60 ml. of conc. ammonia is added (under a protective nitrogen atmosphere). This gives rise to a green color in the nascent suspension, as well as in the benzene. After two hours of stirring, the green benzene solution is decanted (in the absence of air) into a fairly large Schlenk tube (see Part I, p. 75) which has been carefully prepurged with N<sub>2</sub>; it is then dried with solid KOH.

A green crystalline residue remains after vacuum removal of the benzene. This is transferred to a sublimation vessel (complete exclusion of air) and sublimed in high vacuum at 100-105°C.

Yield: with  $(NH_2)_2C(SO_2K)_2$ , 1 g.; with  $(NH_2)_2CSO_2$ , 0.7 g.; or 27 and 20% of theoretical, respectively (based on MoCl<sub>5</sub>).

PROPERTIES:

Green crystals, extremely sensitive to air; decomp. 115°C. Soluble in organic media such as benzene, ether and petroleum ether. Insoluble in water. Very sensitive to oxidation.

REFERENCE:

E. O. Fischer and H. O. Stahl. Chem. Ber. 89, 1805 (1956).

# Molybdenum (II) Chloride

# Mo<sub>3</sub>Cl<sub>6</sub>

 $I_{\bullet} \qquad \qquad 6 \operatorname{MoCl}_{3} = \operatorname{Mo}_{3} \operatorname{Cl}_{6} + 3 \operatorname{MoCl}_{4} \\ 1213.9 \qquad 500.6 \qquad 713.3$ 

Heating of 20 g. of pure  $MoCl_3$  to red heat in a small boat placed in an  $O_2$ -free nitrogen stream gives bright yellow, analytically pure  $Mo_3Cl_6$  (93% yield).

II.  $3 \text{ Mo} + 3 \text{ COCl}_2 = \text{Mo}_3 \text{Cl}_6 + 3 \text{ CO}$ 288.0 296.8 500.6

A stream of  $COCl_2$  (3 bubbles/sec.) is allowed to react with 8 g. of very pure Mo in a Vycor tube (30 minutes at about 610°C). Yield is 90%; 0.5% is lost in the form of side products; the remainder is unreacted Mo. Heating is carried out with a thermostatically controlled electric furnace precalibrated to 610°C.

The sintered reaction mass is finely ground and extracted several times with a mixture of 95 parts of ether and 5 parts of alcohol (reflux). The filtered, golden-yellow solution gives (in vacuum) a scalelike residue which crumbles to a light-yellow dust when ground. This compound corresponds to the formula  $Mo_3Cl_6 \cdot C_2H_5OH$ ; the alcohol cannot be removed without decomposing the product.

PROPERTIES:

Amorphous, dull-yellow powder, stable in air. Infusible, non-volatile;  $d^{2}\frac{5}{4}$  3.714. Insoluble in water, glacial acetic acid, toluene and naphtha. Soluble in alcohols, acetone and pyridine.

**REFERENCES:** 

- I. W. Biltz and C. Fendius. Z. anorg. allg. Chem. <u>172</u>, 384 (1928);
   S. Senderoff and A. Brenner. J. Elektrochem. Soc. <u>101</u>, 28 (1954).
- II. K. Lindner, E. Haller and H. Helwig. Z. anorg. allg. Chem. 130, 210 (1923).

# Molybdenum (III) Chloride

#### MoCl<sub>3</sub>

 $\begin{array}{rrrr} M_{0}Cl_{5} + H_{2} &= M_{0}Cl_{3} + 2 HCl \\ 273.2 & 22.4 l. & 202.3 \end{array}$ 

The reaction tube shown in Fig. 320 is used both for the preparation of  $MoCl_5$  (see p. 1405) and the subsequent reduction to  $MoCl_3$ .



Fig. 320. Preparation of molybdenum (III) chloride. The lengths of the individual tube sections are: a 32 cm., b 8 cm., c 60-75 cm. The I.D. is 2-2.5 cm., 1-1.2 cm. at the constrictions.

Six grams of Mo powder is placed at a, and MoCl<sub>5</sub> is prepared from it (see next preparation); the product is then sublimed into band c by means of a Cl<sub>2</sub> stream. After cooling, the Cl<sub>2</sub> is displaced with CO<sub>2</sub>, and this, in turn, with dry, O<sub>2</sub>-free hydrogen. The left end (near its lowest part) of zone c is now heated to about 250°C, so that a 5- to 10-cm. section of the tube is filled with red vapor. After some time, a white mist of HCl appears at the tube end adjacent to the moisture-retaining silica gel tube. Continued volatilization of the MoCl<sub>5</sub> (which keeps dropping into the lowest section of the tube) results in a copper-red coating.

The heat source is gradually shifted to the right, until all the  $MoCl_5$  is transformed into  $MoCl_3$ . Overheating should be scrupulously avoided. The reaction requires 2-3 hours. At the end, the  $H_2$  is replaced by dry  $CO_2$ , and the remaining  $MoCl_5$  is distilled from section b so that none of it remains in the  $MoCl_3$ . The excess  $MoCl_5$  is driven into adapter d (this tube is attached to the reactor by means of an asbestos seal). After cooling, the reactor tube is cut into several pieces and the crystals are pushed out with a glass rod. Yield: 4-6 g.

PROPERTIES:

Copper- to brown-red powder;  $d^{2\frac{5}{4}}$  3.578. Stable in air; sparingly soluble in pyridine; insoluble in water, alcohol and ether. Forms a blue solution with conc. H<sub>2</sub>SO<sub>4</sub>.

REFERENCES:

H. Biltz and W. Biltz. Übungsbeispiele aus der unorganischen Experimentalchemie [Excercises in Inorganic Experimental Chemistry], 3rd and 4th eds., 1920; W. Biltz and C. Fendius. Z. anorg. allg. Chem. <u>172</u>, 389 (1928); L. P. Liechti and B. Kempe. Liebigs Ann. <u>169</u>, 344 (1873); see also A. Rosenheim, G. Abel and R. Lewy. Z. anorg. allg. Chem. <u>197</u>, 200 (1931).

# Molybdenum (V) Chloride

## MoCl<sub>5</sub>

 $2 \operatorname{Mo} + 5 \operatorname{Cl}_{2} = 2 \operatorname{MoCl}_{5}$ 191.9 110.0 l. 546.5

The apparatus shown in Fig. 321 is used for the chlorination of the Mo. During the experiment, an additional large-diameter piece of glass tubing is attached at c by means of a large-diameter

rubber hose. This glass tube is pointed upward. The left section of reaction tube a-b is charged with 6-10 g. of Mo (for preparation, see p. 1401). Then CO<sub>2</sub> and H<sub>2</sub> (both free of oxygen and very dry) are passed through the tube until the air is completely displaced from a wash bottle which is connected in series. The CO<sub>2</sub> flow is now shut off, and the Mo heated in the hydrogen stream for 1-2 hours (the temperature should be as high as possible). The water (which forms via reduction of the surface oxide layer) is driven off via the open end c by means of a burner and the tube is allowed to cool in the H<sub>2</sub> stream.

The oxide layer may also be reduced in an alternate procedure, whereby the Mo is heated in a dry HCl stream until no further wooly sublimate ( $MoO_3 \cdot 2$  HCl) is formed. The sublimate can be driven into the above mentioned glass tube by gentle warming (use a hood!). It is recommended that a drying tube or a wash bottle with conc.  $H_2SO_4$  be attached at the end of the reaction tube to maintain anhydrous conditions. The outlet gases should pass through the drying arrangement before entering the hood.



Fig. 321. Preparation of molybdenum (V) chloride. Overall length of reaction tube about 1 m.; part a-b 30 cm.; I.D. 2-2.5 cm. and 1-1.2 cm. at the constrictions.

Before chlorination, the tube is cooled, the attached wash bottle (conc.  $H_2SO_4$ ) purged of air with  $Cl_2$ , and the  $H_2$  (or HCl) displaced from the tube with  $Cl_2$ ; during this operation, pinchcock II remains closed. The reaction between Mo and  $Cl_2$  either starts spontaneously or at most needs only very gentle heating with a serial burner for initiation. The reaction is accompanied by the appearance of streams of deep dull-red vapor which condense beyond the constriction b. By gentle heating of the Mo with a serial burner, as well as occasional heating of constriction b by fanning with a Bunsen flame, the MoCl<sub>5</sub> is collected in b-c, where it precipitates in a shower of very fine crystalline leaflets. Intense heating must be avoided. At the end, only a few small gray flakes remain at the left of b. The tube contents are allowed to cool in a  $CO_2$  stream, c is closed off with a cork, and b is sealed off with a torch. The crystals are loosened by tapping, and transferred to a  $CO_2$ -filled, 35-cm.-long storage tube of the same diameter as the combustion tube (the storage tube is slipped over constriction c). The storage tube is then sealed off.

PROPERTIES:

Blue-black, extremely hygroscopic crystals; dark green if oxychloride is present. M.p. 194°, b.p. 268°;  $d^{25}_{4}$  2.927<sub>5</sub>. Soluble in water and alcohol (solvolysis); soluble without decomposition in organic solvents such as ether, CHCl<sub>3</sub>, CCl<sub>4</sub> and CS<sub>2</sub>.

REFERENCES:

H. Biltz and W. Biltz. Übungsbeispiele aus der unorganischen Experimentalchemie [Exercises in Inorganic Experimental Chemistry], 3rd and 4th eds., 1920, p. 216; W. Biltz and A. Voigt. Z. anorg. allg. Chem. <u>133</u>, 299 (1924); P. Liechti and B. Kempe. Liebigs Ann. <u>169</u>, 345 (1873); E. R. Epperson et al. Inorg. Syntheses <u>7</u>, 163 (1963).

# Molybdenum (III) Bromide

#### MoBr<sub>3</sub>

 $2 Mo + 3 Br_2 = 2 MoBr_3$ 191.9 479.5 671.4

Obtained (75% yield) by heating Mo in a stream of  $Br_2$  at 350°C. Separated from by-products by washing in cold, HBr-saturated water.

PROPERTIES:

Formula weight 335.70. Black, densely matted crystalline needles, which decompose to  $Mo_3Br_6$  and  $Br_2$  when calcined in the absence of air. Insoluble in water and acids, readily soluble in boiling anhydrous pyridine, forming  $[MoBr_3py_3]$ .

REFERENCE:

A. Rosenheim, G. Abel and R. Lewy. Z. anorg. allg. Chem. <u>197</u>, 200 (1931).

# Tribromotripyridinemolybdenum

# [MoBr<sub>3</sub>py<sub>3</sub>]

 $\begin{array}{rrrr} MoBr_{3} \ + \ 3 \ py \ = \ [MoBr_{3} py_{3}] \\ 335.7 \ 237.3 \ 573.0 \end{array}$ 

Five grams of  $MoBr_3$  (see preceding preparation) is refluxed at 120°C with 20 g. of anhydrous pyridine (frequent swirling necessary). The hot pyridine solution is rapidly filtered (suction) to remove a small amount of residue and is then treated with conc. HCl until a weak acid reaction is obtained. The brown-yellow precipitate which forms is filtered off and washed with alcohol and ether. For purification the compound is extracted with chloroform in a Soxhlet apparatus. Crystal clusters consisting of small octahedra separate upon slow evaporation of the solution.

PROPERTIES:

Brownish-yellow needles; crystallize in octahedra from chloroform; soluble in pyridine; sparingly soluble in chloroform; insoluble in water, dilute acids, alcohol and ethyl acetate.

REFERENCE:

A. Rosenheim, G. Abel and R. Lewy. Z. anorg. allg. Chem. 197, 201 (1931).

# Potassium Hexachloromolybdate (III)

#### K<sub>3</sub>MoCl<sub>6</sub>

 $2 \operatorname{MoO}_{3} + 6 \operatorname{HCl} + 6 \operatorname{KCl} + 6 \operatorname{e} \rightarrow 2 \operatorname{K}_{3} \operatorname{MoCl}_{6}$   $(\cdot 2 \operatorname{H}_{2} \operatorname{O})$   $360.0 \qquad 447.3 \qquad 852.0$ 

I. A solution of 20 g. of  $H_2MoO_4 \cdot H_2O$  in 150 ml. of conc. HCl and 50 ml. of distilled water is electrolyzed for several hours at about 0.06-0.12 amp./in.<sup>2</sup>; the electrolysis vessel is water cooled and  $CO_2$  is bubbled through the solution. The solution is thus reduced to the red, trivalent state.

Smooth Pt, Hg or amalgamated Pb may be used for the cathode. The carbon anode, immersed in 15% HCl, is separated from the cathodic electrolyte by a clay cell diaphragm.

The reduced solution is evaporated as rapidly as possible over a free flame until its volume is about 90 ml.; it is then saturated with hydrogen chloride and treated with a deaerated 10% solution of 15-20 g. of KCl in distilled water. It is then concentrated at 70°C and reduced pressure until crystals begin to separate, filtered and resaturated with hydrogen chloride while cooling in ice. The crystals are suction-filtered, washed with conc. HCl, with alcoholic HCl and finally with alcohol; they are then dried in vacuum.

II. Potassium molybdate is dissolved in HCl and reduced at a cathode immersed in a clay cell. Gaseous HCl is bubbled through the cathode liquor to precipitate the  $K_3MoCl_6$ .

PROPERTIES:

Formula weight 425.98. d<sup>18</sup> 2.54. Brick-red crystals, readily soluble in water.

**REFERENCES**:

- I. W. R. Bucknall, S. R. Carter and W. Wardlaw. J. Chem. Soc. (London) <u>1927</u>, 513; A. Rosenheim and W. Braun. Z. anorg. Chem. <u>46</u>, 320 (1905).
- II. S. Senderoff and A. Brenner. J. Electrochem. Soc. <u>101</u>, 28 (1954); see also K. H. Lohmann and R. C. Young in: J. C. Bailar, Inorg. Syntheses, Vol. IV, New York-London-Toronto, 1953, p. 97.

# Molybdenum (IV) Oxide

MoO<sub>2</sub>

 $M_0O_3 + H_2 = M_0O_2 + H_2O$ 144.0 22.4 l. 128.0 18.0

Molybdenum (VI) oxide is reduced for 5-7 hours in a stream of  $H_2$  at 450°C; the oxide mixture is then calcined at dark-red heat in a porcelain boat while a stream of HCl is passed over it; this causes any remaining MoO<sub>3</sub> to volatilize as MoO<sub>3</sub>  $\cdot$  2 HCl. Finally the product is allowed to cool under  $H_2$ .

II. MELT REDUCTION OF MoO3 WITH NH3

The following method is useful for preparing pure  $MoO_2$ : Eight grams of dehydrated commercial ammonium molybdate, 7 g. of purified molybdic acid, 14 g. of calcined  $K_2CO_3$ , and 7 g. of  $H_3BO_3$ are fused together in a large-capacity, covered Pt crucible and kept in the molten state for several hours. Cooling yields a

I.

brittle cake, permeated with beautiful  $MoO_2$  crystals; this is easily removed from the crucible. The pure crystals are readily separated by a simple extraction of the melt with boiling water.

III. A mixture of  $MoO_3$  and Mo (2:1 mole ratio) is heated for 40 hours at 700°C in the absence of air.

PROPERTIES:

Brown-violet powder or crystals, insoluble in water;  $d^{2_{4}^{6}}$  4.696. Crystal structure: C 4 (rutile) type.

**REFERENCES:** 

- I. C. Friedheim and M. K. Hoffmann. Ber. dtsch. chem. Ges. <u>35</u>, 792 (1902).
- II. W. Muthmann. Liebigs Ann. 238, 116 (1887).
- III. A. Magnéli, G. Andersson, B. Blomberg and L. Kihlborg. Analyt. Chem. <u>24</u>, 1998 (1952).

# γ-Molybdenum Oxide

M04O11

 $\begin{array}{rrrr} 11 \, MoO_3 \ + \ Mo \ = \ 3 \, Mo_4O_{11} \\ 1583.5 \ & 96.0 \ & 1679.4 \end{array}$ 

Very pure molybdenum powder and sublimed  $MoO_3$  are intimately mixed and charged into a preignited alumina crucible, which is placed at the sealed end of a quartz tube. The mixture is degassed in high vacuum, purged frequently with  $O_2$ -free argon, and then heated for 3 days at 580°C under an argon pressure of 150 mm. The product is allowed to cool, the material is reground to a fine powder, and reheated for 3 days in the same manner.

PROPERTIES:

Formula weight 559.80. Violet crystalline powder. Relatively stable to acids and alkalies. Concentrated  $HNO_3$  causes slow oxidation to  $MoO_3$ . Semiconductor material.  $d_4^{20}$  4.18. Orthorhombic crystals (space group  $D_{2h}^{16}$ ).

REFERENCES:

G. Hägg and A. Magnéli. Ark. Kem. Mineral., Geol. <u>19 A</u>, 1 (1944);
O. Glemser and G. Lutz. Z. anorg. allg. Chem. <u>263</u>, 2 (1950);
A. Magnéli, G. Andersson, B. Blomberg and L. Kihlborg. Analyt. Chem. <u>24</u>, 1998 (1952).

# Lower Molybdenum Hydroxides

#### MOLYBDENUM BLUE, MO<sub>4</sub>O<sub>10</sub>(OH)<sub>2</sub>

Obtained by reaction of nascent hydrogen with  $MoO_3$ .

Fifty ml. of distilled water and 10 ml. of conc. HCl, followed by 3 g. of analytically pure zinc granules, are added to 10 g. of  $MoO_3$ . The mixture is left standing overnight; the blue precipitate is then filtered off, washed until no chloride reaction is evident, and dried over  $P_2O_5$ .

Alternate methods: a) Reduction with  $SnCl_2 \cdot 2 H_2O$  in HCl solution.

b) Synthesis from  $MoO_3$  and Mo powder (O. Glemser and G. Lutz, see below).

Other molybdenum blue compounds: See O. Glemser and G. Lutz (below).

#### PROPERTIES:

Formula weight 477.82. Blue crystalline powder. In air, oxidizes very slowly to  $MoO_3$ . Stable to  $NH_3$  and alkalies. Good electrical conductivity.

REFERENCE:

O. Glemser and G. Lutz. Z. anorg. allg. Chem. 264, 17 (1951).

#### Mo<sub>5</sub>O<sub>5</sub>(OH)<sub>10</sub>

This olive-colored hydroxide is obtained by the action of zinc granules on molybdenum trioxide in conc. HCl.

An Erlenmeyer flask is fitted with a water-filled valve to exclude air (Contat-Göckel attachment), whereupon 1 g. of  $MoO_3$  is charged; 100 ml. of conc. HCl is then added, followed by about 50 g. of zinc granules. The solid phase first becomes blue [formation of  $Mo_4O_{10}(OH)_2$  and  $Mo_2O_4(OH)_2$ ], then red  $[Mo_5O_7(OH)_8]$ , and after about one hour green  $[Mo_5O_5(OH)_{10}]$ . It is desirable to cool the flask with ice during the reduction. The  $Mo_5O_5(OH)_{10}$  is extremely sensitive to air; the latter must therefore be excluded during washing with water. The compound is dried with  $O_2$ -free nitrogen and stored in sealed tubes.

#### PROPERTIES:

Oxidizes instantly in air, evolving heat. Evolves hydrogen on thermal decomposition in vacuum, forming Bordeaux red  $Mo_5O_7(OH)_8$ . This is also the main product obtained in the reaction of aerated water with  $Mo_5O_5(OH)_{10}$ .

REFERENCE:

O. Glemser, G. Lutz and G. Meyer. Z. anorg. allg. Chem. <u>285</u>, 173 (1956).

# Molybdenum (VI) Oxide

## MoO<sub>3</sub>

Prepared from ammonium molybdate and nitric acid.

Boiling conc. nitric acid is added to a boiling solution of pure, recrystallized ammonium molybdate, thus precipitating  $H_2MnO_4$ . After standing for several hours, the granular precipitate is filtered off on a Büchner funnel, washed and dried for 16-20 hours at above 150°C, whereupon it dehydrates to  $MoO_3$ . The  $MoO_3$  may be purified by sublimation in a quartz tube at 780°C.

SYNONYMS:

Molybdenum trioxide; molybdic anhydride.

PROPERTIES:

Formula weight 143.95. White powder, turning yellow on heating and reverting to white on cooling. The sublimed product consists of sparkling, colorless crystalline flakes. M.p. 795°C, b.p. 1155°C;  $d^{26}_{4}$  4.696. Solubility at 28°C: 0.490 g. MoO<sub>3</sub>/liter H<sub>2</sub>O. Crystal form: rhombic. Space group  $V_{h}^{16}$ .

REFERENCES:

W. C. Schumb and W. H. Hartford. J. Amer. Chem. Soc. <u>56</u>, 2613 (1934);
 O. Hönigschmid and G. Wittmann. Z. anorg. allg. Chem. <u>229</u>, 66 (1936).

# Molybdic Acid

## $H_2M_0O_4 \cdot H_2O$

Prepared from ammonium molybdate and nitric acid.

An aqueous solution of ammonium molybdate (150 g./liter) is poured into 1 liter of 30% nitric acid [300 ml. of conc. HNO<sub>3</sub> (d 1.42) per liter] at room temperature (vigorous stirring). Then, 200 g. of solid NH<sub>4</sub>NO<sub>3</sub> is dissolved in the clear liquid, which is then seeded with a few granules of  $H_2MOO_4 \cdot H_2O$ . On standing for  $8\text{-}10~\text{days}~\text{H}_2\text{MoO}_4$   $\cdot$  H\_2O separates in almost theoretical yield. The acid is washed for several days by repeated decantation with ice water.

PROPERTIES:

Formula weight 179.98. Transparent canary-yellow crystals, very sparingly soluble in water. Loses 1 mole of  $H_2O$  by standing for about 2 weeks in vacuum over  $H_2SO_4$ . d<sup>15</sup> 3.124. Monoclinic-prismatic crystals.

REFERENCE:

A. Rosenheim. Z. anorg. Chem. 50, 320 (1906).

# Ammonium Oxopentachloromolybdate (V) (NH4)2[M0OCl5]

A) PREPARATION OF MoO3 SOLUTION

The  $MoO_3$  (100 g.) is dissolved in HCl (d 1.16, 500 ml.) by heating. The solution is concentrated to 250 ml., filtered and made up to 500 ml. with HCl.

# B) ELECTROLYTIC REDUCTION

The above-prepared solution (75 ml.) is diluted with an equal volume of  $H_2O$  and electrolyzed at a platinized Pt cathode (surface = 5 cm.<sup>2</sup>) at 2.5 amp. until hydrogen evolves. The anode is made of smooth Pt sheet; it is immersed in 5 N HCl, which is separated from the cathode space by means of a clay cell.

The resulting red-brown solution is vacuum-concentrated to 50 ml. and treated with a solution of 9 g, of  $NH_4Cl$  in 30 ml of water; the mixture is heated for about one minute.

Hydrogen chloride is introduced while cooling the flask and emerald-green crystals are precipitated. These are recrystallized by dissolving in a minimum quantity of water at 80°C and saturating the solution with HCl while cooling. The crystals are washed with conc. HCl and dried in vacuum over KOH.

PROPERTIES:

Formula weight 325.32. Emerald-green octahedral crystals, which dissolve in water with hydrolysis and development of brown color.

REFERENCES:

R. G. James and W. Wardlaw. J. Chem. Soc. (London) <u>1927</u>, 2146;
 F. Foerster and R. Fricke. Z. angew. Chem. <u>36</u>, 458 (1923).

# Potassium Hydrogen Diperoxomonomolybdate

## KHMoO<sub>6</sub> · 2 H<sub>2</sub>O

Potassium molybdate solutions rich in  $H_2O_2$  are treated with one equivalent of mineral acid per mole of molybdate, whereupon KHMoO<sub>6</sub> · 2  $H_2O$  crystallizes.

PROPERTIES:

Formula weight 268.09. Long, pale yellow, crystalline needles.

REFERENCES:

K. F. Jahr. Ber. Ges. Freunde TH. Berlin <u>1939</u>, 91; K. F. Jahr. Naturforsch. u. Medizin in Deutschland [Scientific Research and Medicine in Germany] 1939-1946 (FIAT Review) <u>25</u>, III, 189.

# Tetraamminezinc Tetraperoxomolybdate (VI)

# [Zn(NH<sub>3</sub>)<sub>4</sub>]MoO<sub>8</sub>

Prepared from ammonium molybdate,  $NH_4OH$  and  $ZnSO_4$  by addition of  $H_2O_2$ .

A mixture of 100 ml. of water, 100 ml. of conc. ammonia (d 0.91), and 20 ml. of an ammonium molybdate solution containing 1 g.-atom of Mo per liter is cooled to  $-12^{\circ}$ C in an ice-salt mixture. Lower temperatures cannot be used, since water is frozen out (ice) at  $-14^{\circ}$ C. Then 30 ml. of Perhydrol (30% H<sub>2</sub>O<sub>2</sub>) is added, followed by 20 ml. of 1 M ZnSO<sub>4</sub> solution (brief but thorough stirring). The solution is allowed to stand undisturbed for 1.5 hours at  $-12^{\circ}$ C. After one half hour it is examined for signs of incipient crystallization. If none is observed, crystal formation is induced by careful rubbing of the flask walls with a glass rod. The crystals are filtered off, washed twice with ice-cold 96% alcohol and then twice with ice-cold ether. Yield: about 3 g.

## PROPERTIES:

Formula weight 357.46. Deep red-brown, lustrous crystals; soluble in water (decomposition). Ammoniacal solutions are more

stable. Insoluble in organic solvents. Appreciably more stable than the corresponding potassium salt and barely explosive. However, it is not advisable to seal it into ampoules. May be stored in a desiccator over KOH and under  $NH_3$  for about 1 week.

**REFERENCE:** 

K. Gleu. Z. anorg. allg. Chem. 204, 73 (1932).

# Molybdenum (IV) Sulfide

MoS<sub>2</sub>

I.

$MoO_2$	+	3S =	$MoS_2$	+	SO <sub>2</sub>
128.0		96.2	160.1		

A mixture of 150 g. of  $K_2CO_3$ , 310 g. of S, and 200 g. of  $MoO_2$  is heated at red heat for one half hour. After cooling and extracting with water, the residue is 80 g. of  $MoS_2$ . It is also possible to start with 200 g. of ammonium molybdate, 150 g. of  $K_2CO_3$ , and 280 g. of S; this affords a more crystalline product, although in lower yield. According to Bell and Herfert, one may also start with  $MoO_3$ ,  $K_2CO_3$  and S.

II.

 $Mo + 2S = MoS_2$ 96.0 64.1 160.1

Stoichiometric quantities of Mo and S are heated in an iron tube. The  $MoS_2$  thus obtained has the crystal lattice of natural molybdenite.

SYNONYM:

Molybdenum disulfide.

PROPERTIES:

Opaque, gray-blue leaflets with a greasy feel, or graphite-like powder. Sublimes at 450°C;  $d^{15}$  5.06. Soluble (decomposition) in aqua regia. Decomposed by  $H_2SO_4$  to  $MOO_3$ . Electrical conductor whose conductivity increases with illumination. Diamagnetic. Crystal structure: C 7 type.

REFERENCES:

- I. M. Guichard, Comptes Rendus Hebd. Séances Acad. Sci. <u>129</u>, 1239 (1899); Ann. Chim. Phys. 7, <u>23</u>, 552 (1901); R. E. Bell and R. E. Herfert. J. Amer. Chem. Soc. 79, 3351 (1957).
- II. A. E. van Arkel. Recueil Trav. Chim. Pays Bas 45, 442 (1926).

# Ammonium Tetrathiomolybdate

# (NH<sub>4</sub>)<sub>2</sub>MoS<sub>4</sub>

Prepared by treating an ammoniacal ammonium molybdate solution with  $H_2S$ .

A solution of 5 g. of  $(NH_4)_6Mo_7O_{24} \cdot 4 H_2O$  in 15 ml. of water is prepared and treated with 50 ml. of ammonia (d 0.94). Then  $H_2S$  is introduced. The solution first turns yellow, later deep red, and after half an hour a copious quantity of crystals, some of them well-formed, precipitates suddenly. The crystals are washed with cold water, then with alcohol, and dried in vacuum.

PROPERTIES:

Formula weight 260.27. Blood-red crystals with metallic surface luster, readily soluble in water, very sparingly soluble in alcohol.

REFERENCE:

G. Krüss. Liebigs Ann. 225, 29 (1884).

# Potassium Octacyanomolybdate (IV)

# $K_4[Mo(CN)_8] \cdot 2 H_2O$

I. A solution of Mo (III) obtained by electrolytic reduction is oxidized to the quadrivalent state with a stoichiometric quantity of  $MoO_3$  and is then treated with  $NH_4SCN$  and pyridine. The precipitate is converted to  $K_4[Mo(CN)_8] \cdot 2 H_2Oby$  reaction with KCN.

A solution of 15 g. of pure  $MoO_3$  in 150 ml. of 8 N HCl is reduced to the trivalent state at a lead cathode (5-10 amp.) while cooling the cathode cell (porous clay) with water and bubbling  $CO_2$  through the solution. A graphite rod immersed in 15% HCl serves as the anode.

The reduced solution is treated with a solution of 7.5 g. of  $MoO_3$  in 75 ml. of 8 N HCl; the solution thereby changes from red to green. The Mo (IV) salt solution thus obtained is poured into a concentrated solution of 95 g. of NH<sub>4</sub>SCN; 60 g. of pyridine is added, and the solution is made weakly alkaline with ammonia. These reactions are best carried out under  $CO_2$  in the absence of air.

A black oil separates during the neutralization; this solidifies to a solid black tar on cooling with ice. The supernatant liquid is decanted; the tar is washed with water and added slowly to a hot, saturated 10 M KCN solution. The reaction mixture is concentrated until crystals begin to deposit; it is then cooled in ice and filtered with suction.

The mother liquor is further concentrated and the cyano complex is precipitated with alcohol (rapid stirring). The crude product is dissolved in some water, the solution heated for some time with activated charcoal, filtered, concentrated by evaporation, and finally treated with alcohol. The precipitate of  $K_4[Mo(CN)_8] \cdot$  $2 H_2O$  is fairly pure except for traces of thiocyanate, which can be removed by an additional recrystallization. Yield: about 55 g.

Alternate methods:

II. The product of the reaction of  $MoO_3$  with HSCN and pyridine is treated with KCN. The yield is low. (See original reference for details.)

III. Via the following intermediates:

$$\begin{array}{cccc} \mathrm{HCl},\mathrm{H}_{2}\mathrm{O} & \mathrm{KSCN} & \mathrm{pyridine} \\ \mathrm{MoO}_{3} & \rightarrow & \mathrm{H}_{2}\mathrm{MoO}_{4} & \rightarrow & \mathrm{Mo(OH)}_{2}(\mathrm{SCN})_{3} & \rightarrow \\ & & & \mathrm{KCN} \\ & & \mathrm{Mo(OH)}_{2}(\mathrm{SCN})_{3} \cdot 2 \ \mathrm{py} & \rightarrow & \mathrm{K}_{4}[\mathrm{Mo(CN)}_{8}] \cdot 2 \ \mathrm{H}_{2}\mathrm{O} \end{array}$$

PROPERTIES:

Formula weight 496.50 (dihydrate). Golden or bright-yellow tablet-shaped crystals (rhombic bipyramidal); gives up its water of crystallization at 105-110°C. Very readily soluble in water, insoluble in ether. Solubility in absolute alcohol at 20°C: 0.017 g. per liter.  $d^{25}_{4}$  (anhydrous salt) 2.337.

REFERENCES:

- I. H. H. Willard and R. C. Thielke. J. Amer. Chem. Soc. <u>57</u>, 2610 (1935).
- II. A. Rosenheim. Z. anorg. Chem. <u>54</u>, 97 (1907); see also W. Biltz, E. Eschweiler and A. Bodensiek. Z. anorg. allg. Chem. <u>170</u>, 168 (1928).
- III. N. H. Furman and C. O. Miller in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 160.

## Tungsten

W

$$WO_3 + 3H_2 = W + 3H_2O$$
  
231.9 67.3 l. 183.9 54.1

Pure ignited  $WO_3$  is placed in a porcelain or, better, a nickel boat, which is then inserted into a tube of unglazed porcelain or

other refractory material. The tube is heated by means of an electric furnace while passing through a stream of pure, dry H<sub>2</sub>. The initial temperature is 800°C (maintained for some time), later 1000-1200°C. The reduction proceeds fairly rapidly and is complete when no further H<sub>2</sub>O vapor evolves. The tube is allowed to cool in the H<sub>2</sub> stream. Gray metal powder is the product.

Other authors report that the reduction with very pure  $H_2$  may even be completed at 800 °C (8 hours of heating).

The particle size of the metal powder does not depend on that of the starting WO<sub>3</sub>, but (principally) on the reaction temperature, as well as the heating time and the H<sub>2</sub> flow rate. An especially fine tungsten powder is obtained when the above directions are followed; a very coarse powder results from reduction with moist H<sub>2</sub> above 1500°C. The product is pure if the starting WO<sub>3</sub> is also pure.

Alternate method:

$$WO_3 + 3Zn = W + 3ZnO$$
  
231.9 196.1 183.9 244.1

Pure, freshly ignited WO<sub>3</sub> is cooled and mixed with 1.5 times the theoretical quantity of dry Zn dust. This mixture is compressed in a crucible and overlaid with another 3- to 5-mm. layer of Zn dust, which is similarly compressed. The crucible is then closed off with a well-fitting cover and heated briefly to red heat. A vigorous reaction ensues after about 5 minutes, as shown by a cloud of evolving ZnO. The crucible is then allowed to cool completely while still covered. The light-gray top layer, composed principally of ZnO, is removed. The black crucible contents are crushed and thoroughly boiled with dil. HCl to remove ZnO and any Zn still present. The black residue is allowed to settle for a moment and the acid is decanted. The metal powder is washed by decantation with water once or twice, filtered rapidly with suction, and rewashed with water. In this operation the powder must always be covered with liquid. Finally the metal is washed with alcohol, suction-filtered, and dried in air. Careful preparation should yield a black powder containing at least 99.8% W.

SYNONYM:

Wolfram.

PROPERTIES:

Tungsten powder is gray to black, depending on particle size; the solid exhibits a light gray, lustrous surface. M.p.  $\sim 3650$  °C, b.p. > 5000 °C (calculated from the vapor pressure curve); d 19.3. Hardness 4.5-8, depending on the history. Crystal structure: A 2 type.

**REFERENCES**:

H. Funk. Darstellung der Metalle im Laboratorium [Preparation of Metals in the Laboratory], Stuttgart, 1938, p. 72 f.; O. Ruff. Angew. Chem. <u>25</u>, 1892 (1912); O. Glemser and H. Sauer. Z. anorg. allg. Chem. <u>252</u>, 145 (1943); M. Delépine. Comptes Rendus Hebd. Séances Acad. Sci. <u>131</u>, 184 (1900); L. Weiss and A. Martin. Z. anorg. allg. Chem. <u>65</u>, 308 (1910).

## Tungsten (V) Chloride

#### WCl₅

 $WCl_6 + \frac{1}{2}H_2 = WCl_5 + HCl$ 396.7 11.2 l. 361.2

Reduction of WCl<sub>5</sub> vapor with  $H_2$  for too short a period of heating and at too low a temperature results in hexachloride-containing WCl<sub>5</sub>. Too high a temperature affects the yield adversely, since considerable amounts of lower chlorides form.

A tube of high-melting glass (Fig. 322) is used. The WCl<sub>6</sub> produced in section *a* is distilled into the 50-cm.-long section *b* in a stream of  $H_2$ . Section *b* is heated to the reduction temperature, which lies somewhat above the boiling point of WCl<sub>6</sub> (350-400 °C, electric furnace). The WCl<sub>5</sub> is distilled into storage tube *c* in a stream of N<sub>2</sub>, thereby separating it from the lower chlorides; tube *c* is then sealed off at both ends.



Fig. 322. Preparation of tungsten (V) chloride.

PROPERTIES:

Black, crystalline solid with somewhat greenish luster. Extremely hygroscopic. M.p. 248°C;  $d_{4}^{25}$  3.875. Water causes immediate decomposition; somewhat soluble in dry CS<sub>2</sub>.

REFERENCES:

H. E. Roscoe. Liebigs Ann. <u>162</u>, 356 (1872); W. Biltz and A. Voigt.
Z. anorg. allg. Chem. <u>133</u>, 301 (1924); W. Biltz and C. Fendius.
Ibid. <u>172</u>, 385 (1928); W. Klemm and H. Steinberg. Ibid. <u>227</u>, 193 (1936).

# Tungsten (VI) Chloride

#### WCl<sub>6</sub>

I.

# $W + 3 Cl_2 = WCl_6$ 183.9 66.0 l. 396.7

Prepared in a Vycor tube with several constrictions (Fig. 323). A quartz tube is even better and the special apparatus described by Hönigschmid and Menn is the best.

Fig. 323. Preparation of tungsten (VI) chloride.

A procelain or quartz boat containing W powder is placed in section *a* and is heated for 1-2 hours in a stream of  $H_2$  (700-1000°C). After cooling, the  $H_2$  flow is discontinued, the  $H_2$  displaced by  $N_2$ ; after about 30 minutes, air-free Cl<sub>2</sub> (see p. 272 for preparation) is introduced.

The tube area containing the boat is gradually heated to 600 °C. The first product is a small amount of red oxychloride; later products consist only of blue-black hexachloride, which deposits in section *a* (beyond the heated boat) in the form of sparkling crystals. The tube constrictions are kept at 350-400 °C during the chlorination, which requires 2-3 hours. Next the red oxychloride forerun is driven into *e*; then the WCl<sub>5</sub> collected in *a* is sublimed into *b* at 350-400 °C, while an additional forerun deposits in *c*. For further purification, the WCl<sub>6</sub> can be sublimed into *c* (the oxychloride is moved into *d* prior to this sublimation). Finally, the WCl<sub>6</sub> is fused in a stream of Cl<sub>2</sub>; on resolidification, the mass bursts into small crystals with a loud crackling noise. After cooling, the Cl<sub>2</sub> is displaced with dry, O<sub>2</sub>-free nitrogen and the tube section containing the purified WCl<sub>6</sub> is sealed off at both ends.

II.  $WO_3 + 3 CCl_4 = WCl_6 + 3 COCl_2$ 231.9 461.5 396.7 296.8

The chief requirements in this method are the presence of an excess of  $CCl_4$ , completely anhydrous conditions, and thorough completion of the reaction; if these conditions are not observed, red by-product WOCl<sub>4</sub> forms readily. Moreover, the WOCl<sub>4</sub> has the undesirable property of catalyzing the hydrolysis of WCl<sub>6</sub> in moist air.

A dry glass bomb tube about 50 cm. long is charged with 0.5 g. of WO<sub>3</sub> and 11 g. of CCl<sub>4</sub> (the WO<sub>3</sub> must be completely dehydrated by previous ignition, after which it should be pure yellow; the CCl<sub>4</sub>

is predried by long standing over granular  $CaCl_2$  or  $P_2O_5$  and is saturated with Cl<sub>2</sub>). The tube is then sealed. The water vapor from the torch flame should not be allowed to enter the tube. The tube is placed in a protective iron jacket and slowly heated (1.5-2 hours) to about 450°C; it is kept at this temperature for 7-8 hours. After slow cooling, the tube is very carefully transferred to a welldrawing hood without removing it from its protective iron tube. The latter is inclined and clamped to a support. The sealed tip of the glass tube is then heated at low hood vacuum until the phosgene, present in the tube under high pressure, blows the tip away. The phosgene is driven out by vacuum as far possible, the CCl4 poured off, the residual solid washed once with fresh CCl<sub>4</sub>, and the reactor tube connected to an aspirator via a CaCl<sub>2</sub> drying tower. All the CCl<sub>4</sub> is volatilized and the tube is resealed since the beautiful, almost black crystals of WCl<sub>e</sub> hydrolyze in air to yellowred WOCl<sub>4</sub> and yellow WO<sub>2</sub>Cl<sub>2</sub>. Proper procedure yields an almost black solid with no red or yellow spots.

PROPERTIES:

Blue-black, moisture-sensitive crystals. M.p. 275°C, b.p. 347°C; d<sup>25</sup>/<sub>4</sub> 3.520. Indefinitely stable if stored in a dark desiccator over H<sub>2</sub>SO<sub>4</sub>. Very slightly soluble but decomposed in water; the purer the WCl<sub>6</sub>, the lower the decomposition rate. Very readily soluble in alcohol (with yellow color), CHCl<sub>3</sub>, CCl<sub>4</sub> (with red and dark-brown color, respectively), CS<sub>2</sub>, ether, benzene, ligroin and acetone. These solutions decompose on long standing in air, and very rapidly on heating or addition of water. Good crystals are obtained by heating WCl<sub>6</sub> in CCl<sub>4</sub> to 100°C in a sealed tube, followed by slow cooling (rectangular tablets and four-sided prisms). Crystallizes in space group C<sup>2</sup><sub>31</sub>.

REFERENCES:

- I. O. Hönigschmid and W. Menn. Z. anorg. allg. Chem. <u>229</u>, 58 (1936); see also M. H. Lietzke and M. L. Holt in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 163.
- II. W. Jander. Lehrbuch für das anorganisch-chemische Praktikum [Lab. Text for Inorg. Chemistry], 5thed., Leipzig, 1944, p. 403.

# Tungsten (IV) Oxide

#### WO<sub>2</sub>

 $WO_3 + H_2 = WO_2 + H_2O_{231,9} + 22.4 l, 215.9 + 18.0$ 

Stable at 900°C in a gaseous atmosphere composed of 40-55% H<sub>2</sub> and 45-60% H<sub>2</sub>O.

I. Pure  $H_2$  is subjected to an additional purification over silica gel and is then slowly passed through a water-filled flask to saturate it with water vapor. The flask is held in an 85°C thermostat. To avoid condensation of the water vapor thus taken up, the tube which connects the flask to the reactor is wrapped with electric heating tape and heated to about 100°C. The  $H_2/H_2O$ mixture then flows over a boat with WO<sub>2</sub> set in a porcelain or a quartz reactor tube surrounded by a tubular electric furnace and heated to 800-900°C. The reduction is complete in 2 hours. The product is allowed to cool in an O<sub>2</sub>-free nitrogen stream. The nitrogen is admitted through a 3-way stopcock located between the water flask and the reactor.

II. A mixture of  $WO_3$  and W (corresponding to the formula  $WO_{2.00}$ ) is heated for 40 hours at 950°C in a small evacuated and sealed quartz tube.

PROPERTIES:

Brown crystalline powder. M.p. 1500-1600°C under N<sub>2</sub>, b.p. 1730°C; appreciably volatile above 1050°C;  $d^{25}_{4}$  11.05. Hardness 5-5.5. Crystal structure: C 4 (rutile) type.

**REFERENCES**:

- I. O. Glemser and H. Sauer. Z. anorg. allg. Chem. <u>252</u>, 151 (1943);
   J. A. M. van Liempt. Ibid. <u>126</u>, 184 (1923); L. Wöhler and
   R. Günther. Z. Elektrochem. 29, 281 (1923).
- II. A. Magnéli, G. Andersson, B. Blomberg and L. Kihlborg. Analyt. Chem. 24, 1998 (1952).

# y -Tungsten Oxide

#### W<sub>18</sub>O<sub>49</sub>

Very pure W powder and very pure  $WO_3$  are intimately mixed in the prescribed ratio of  $WO_{2.72}$  and heated for 6 hours at 800°C in a small evacuated and sealed quartz tube. The product is ground and treated again for 24 hours at 800°C in the same manner.

#### PROPERTIES:

Formula weight 4853.47. Red-violet crystalline powder; semiconductor.  $d_{4}^{20}$  7.72. Deformed D0<sub>9</sub> type.

REFERENCES:

O. Glemser and H. Sauer. Z. anorg. allg. Chem. <u>252</u>, 144 (1943);
 A. Magnéli. Ark. Kem. <u>1</u>, 223 (1949); A. Magnéli, G. Andersson, B. Blomberg and L. Kihlborg. Anal. Chem. 24, 1998 (1952).

#### Tungsten Blue

#### $H_{0.5}WO_3$

Produced by reaction of nascent hydrogen with WO<sub>3</sub>.

Fine  $WO_3$  powder is slurried with distilled water in an Erlenmeyer flask, conc. HCl and analytically pure Zn granules are added, and the flask is closed with a valve which excludes air (Contat-Göckel attachment). When the Zn is consumed, the supernatant liquid is rapidly decanted and fresh conc. HCl and Zn granules are added. This is repeated until the reaction product is brown. Washing, drying and transfer of the product to a storage vessel must be carried out in the absence of oxygen.

PROPERTIES:

Formula weight 232.42. Brown to violet powder;  $d_{4}^{20}$  7.35. Very readily oxidized. Evolves  $H_2$  along with  $H_2O$  on thermal decomposition. Oxidized by water with  $H_2$  evolution. Slow oxidation affords blue  $H_{0.33}WO_3$  and  $H_{0.1}WO_3$ .  $DO_9$  type with tetragonal distortion.

**REFERENCE:** 

O. Glemser and C. Naumann. Z. anorg. allg. Chem. <u>265</u>, 288 (1951).

## Tungsten (VI) Oxide

#### WO<sub>3</sub>

L  $Na_2WO_4 + 2 HCl = WO_3 + H_2O + 2 NaCl$  $(\cdot 2 H_2O)$ 330.0 72.9 231.9

The yellow  $WO_3$  is obtained by slow dropwise addition of a warm, saturated solution of  $Na_2WO_4$  to 2-3 times its volume of boiling conc. HCl, followed by additional heating (for 1 hour) on a steam bath. The precipitate is allowed to settle, washed with

5%  $NH_4NO_3$  solution until no further Cl<sup>-</sup> reaction is obtained, and dried, first at 120°C and finally at 600°C.

PROPERTIES:

Lemon-yellow powder. M.p. about 1470°C, b.p. about 1700°C;  $d_{4}^{25}$  7.27. Crystals triclinic, pseudomonoclinic. Space group C<sub>i</sub>.

REFERENCE:

W. Reinders and A. W. Vervloet. Receuil Trav. Chim. Pay-Bas <u>42</u>, 627 (1923).

# Yellow Tungstic Acid

# H<sub>2</sub>WO<sub>4</sub>

I.

 $\begin{array}{rcl} CaWO_4 \ + \ 2 \ HX \ = \ H_2WO_4 \ + \ CaX_2 \\ 288.0 & 249.9 \end{array}$ 

A boiling mixture of 50 ml. of  $H_2O$ , 40 ml. of conc. HCl, and 40 ml. of HNO<sub>3</sub> is treated with 20 g. of pure CaWO<sub>4</sub>. The resulting yellow precipitate is washed 8 times by decantation with slightly acidified water, and dissolved in 50 ml. of conc. ammonia. The clear filtrate is heated to boiling and treated with acid (60 ml. of  $H_2O$ , 50 ml. of HNO<sub>3</sub> and 10 ml. of HCl) to precipitate yellow tungstic acid, which is washed several times by decantation with pure  $H_2O$ , filtered through a leaf filter and slurried in pure  $H_2O$ . The suspension settles on standing for 14 days, during which an electric current is occasionally passed through it (Pt electrodes); the clear supernatant liquid is siphoned off; the residue is concentrated on a steam bath and then thoroughly dried in a desiccator over solid NaOH. The tungstic acid product is free of HCl and has the composition WO<sub>3</sub>  $\cdot$  1.13  $H_2O$ .

II. A boiling solution of 200 g. of ammonium tungstate [composition 2 (NH<sub>4</sub>)<sub>2</sub>O, 0.5 WO<sub>3</sub>, 3 H<sub>2</sub>O] in 4.48 liters of H<sub>2</sub>O is poured into 2 liters of boiling 35.4% HCl. The deep-yellow precipitate is filtered off and purified by a nine-day dialysis (until the wash water is free of Cl<sup>-</sup> and the pH has reached a constant value of 4.4). Air drying of the residue gives a 55% yield of tungstic acid of composition WO<sub>3</sub> · 1.18 H<sub>2</sub>O; it still contains traces of NH<sub>3</sub> and Cl<sup>-</sup>.

PROPERTIES:

Yellow powder, which appears amorphous under the microscope; it is claimed that its x-ray pattern is crystalline.

REFERENCES:

I. G. F. Hüttig and B. Kurbe. Z. anorg. allg. Chem. <u>122</u>, 45 (1922). II. A. M. Morley. J. Chem. Soc (London) <u>1930</u>, 1990.

## Tungsten Oxytetrachloride

#### WOCl<sub>4</sub>

 $WO_3 + 2 SOCl_2 = WOCl_4 + 2 SO_2$ 

I. A mixture of WO<sub>3</sub> with 4 times its weight of SOCl<sub>2</sub> is heated in a sealed tube for 6-12 hours at 200°C. The reaction proceeds to completion only if the SO<sub>2</sub> is vented by opening the sealed tube for a brief time. The red WOCl<sub>4</sub> crystallizes from the excess SOCl<sub>2</sub> in long, well-formed needles, and is purified by removing the SOCl<sub>2</sub> in vacuum.

II. A sealed tube is used to heat the WO<sub>3</sub> with a solution of  $Cl_2$  in  $CCl_4$  (3 hours at 200°C). The tube is cooled and opened (caution: phosgene is present!); the WOCl<sub>4</sub> is washed in the tube with dry  $CCl_4$ ; and the tube and its contents are heated to 160°C and immediately placed in a desiccator.

PROPERTIES:

Long, lustrous, red needles, yellow in transmitted light. M.p. 209°C, b.p. 232°C. Decomposed at once by water, more slowly by atmospheric moisture, forming tungstic acid.

**REFERENCES**:

- I. H. Hecht, G. Jander and H. Schlapmann. Z. anorg. allg. Chem. 254, 261 (1947).
- II. A. Michael and A. Murphy. Amer. Chem. J. 44, 382 (1910).

# Tungsten (IV) Sulfide

#### WS<sub>2</sub>

$$W + 2S = WS_2$$
  
183.9 64.1 248.0

A stoichiometric mixture of W and S is heated at 800 °C under very pure  $N_2$  in a sealed quartz tube for 24 hours.

I.

II.  $2 WO_3 + 7 S = 2 WS_2 + 3 SO_2$ 463.8 224.4 496.1 192.2

An intimate mixture of 33 g. of  $WO_3$ , 40 g. of sulfur and 15 g. of  $K_2CO_3$  is placed in a tubular crucible of unglazed porcelain (190 mm. long, 35 mm. I.D.) which is closed off with a perforated asbestos lid. The crucible is heated in a vertical tubular furnace at 600-700°C until the reaction is complete and the excess S has burned off. It is then heated for an additional 15 hours at 1400°C while  $H_2S$  is passed through. Large crystals of pure sublimed WS<sub>2</sub> are thus obtained.

Finely crystalline  $WS_2$  is obtained from 92 g. of W powder and 35 g. of sulfur by following the above directions but heating only for 7 hours at 1450°C in a stream of  $H_2S$ .

PROPERTIES:

Blue-gray crystals with a metallic luster, insoluble in water;  $d^{10}_{0}$  7.5. Crystal structure: C 7 type. Exhibits catalytic and radio-detector properties.

**REFERENCES**:

- I. O. Glemser, H. Sauer and P. König. Z. anorg. Chem. <u>257</u>, 241 (1948).
- II. E. Tiede and H. Lemke. Ber. dtsch. chem. Ges. 71, 584 (1938).

# Tungsten Hexaphenoxide

#### W(OC<sub>6</sub>H<sub>5</sub>)<sub>6</sub>

$$WCl_6 + 6 HOC_6H_5 = W(OC_6H_5)_6 + 6 HCl$$
  
396.7 554.7 955.3

One wt. part of  $WCl_6$  and about 10 parts of phenol are heated in a long, large test tube. Vigorous evolution of HCl begins as soon as the phenol melts. The tube is now heated over a small flame until the phenol boils. After some time the melt (which initially is brown-black even in thin layers) becomes deep red. Boiling is continued for a short time; the tube is then cooled while rotating it to distribute the melt on the walls of the vessel. The cooled melt is treated with some alcohol while crushing with a glass rod. The excess phenol dissolves in the alcohol and the product separates as a brick-red powder. It is filtered off with suction, washed with alcohol, and recrystallized from the latter. SYNONYM:

Tungsten hexaphenolate.

PROPERTIES:

Dark-red needles or leaflets. M.p. 98°C. Readily soluble in  $CCl_4$ ,  $CS_2$ ,  $C_6H_6$ , etc. Relatively poorly soluble in cold alcohol.

REFERENCE:

H. Funk and W. Baumann. Z. anorg. allg. Chem. 231, 265 (1937).

# Potassium Enneachloroditungstate (III)

#### K<sub>3</sub>W<sub>2</sub>Cl<sub>9</sub>

Prepared by electrolytic reduction of a KCl-containing solution of  $WO_3$  in conc. hydrochloric acid.

A solution of 10 g. of  $WO_3 \cdot H_2O$  in a conc. solution of 7.5 g. of  $K_2CO_3$  is prepared. This solution (volume of about 15 ml.) is added in 2 or 3 portions to 260 ml. of conc. HCl (40°C); the hydrate, which precipitates after each addition, is allowed to redissolve before the next portion is introduced. After complete solution is finally obtained, the liquid is quickly cooled to 0°C. The crystalline precipitate thus obtained is composed for the most part of KCl. The solution, filtered through a fritted glass funnel, contains 3-4% WO<sub>3</sub> as H[WO<sub>2</sub>Cl<sub>3</sub>], and is used in electrolysis.

The cathode vessel is a porous clay cell of about 6.5 cm. I.D. and 21 cm. high; this is set in a heavy-wall 12.5-cm.-I.D. and 20-cm,-high glass cylinder and centered with the aid of a rubber stopper. The rubber stopper contains five holes: two for the symmetrically placed carbon anodes, two for glass tubes used for passing CO<sub>2</sub> through the anodic electrolyte (so as to decrease the Cl<sub>2</sub> concentration in the latter), and a fifth hole through which the electrolyte is introduced. A drain is provided at the bottom of the anode vessel to permit rapid emptying at the end of the run. The clay cell is closed off hermetically with a rubber stopper, which carries a gas-tight stirrer (Hg seal), the cathode lead, and an opening for removal of samples during the electrolysis. The anodic liquor is conc. hydrochloric acid, while the clay cell is charged with 450 ml. of the W (VI) solution described above. The latter is reduced at 40°C and a current density of 0.4 amp./in.<sup>2</sup> until it turns yellow-green and the consumption of permanganate becomes constant.

The cathode is a 140-cm.<sup>2</sup> Pb sheet, which may be amalgamated if necessary. Runs with amalgamated electrodes require a longer

time (up to 2 hours) than do those with pure Pb cathodes (80-90 min.), but the product solution is free of Pb.

To obtain reproducible results with Pb cathodes, the latter must be formed by alternating anodic and cathodic polarization in 2 N  $H_2SO_4$ ; this is unnecessary with the amalgamated electrodes (because of the purely chemical reducing action of the cathode metal, it is best to admit the cathodic liquid only after the voltage has been applied). The Pb, which initially goes into solution and later reprecipitates, causes no problem either during electrolysis or in the later workup of the solution.

After the completion of the reduction, the cathode is carefully removed from the solution, the anode vessel rapidly emptied, and the cathodic liquid containing the precipitated crystals poured into an Erlenmeyer flask, where it is then saturated with HCl while chilling in ice-salt mixture. After 1-2 days, the crystalline precipitate is collected on a fritted-glass funnel, washed with some conc. hydrochloric acid, then with alcohol and ether, and dried in a stream of air. The yield is about 60% based on the  $WO_3 \cdot H_2O$  used.

For purification,  $K_3W_2Cl_9$  is reprecipitated with a readily soluble potassium salt. For example, 25 g. of  $K_3W_2Cl_9$  is dissolved in 175 ml. of boiled cold water; the solution is filtered into an Erlenmeyer flask which contains 150-175 g. of solid KSCN. The flask is then shaken. A copious quantity of deep green to yellow-green crystals separates even during the filtration; the solution meanwhile turns red. After one hour of standing in the cold, the crystals are suction-filtered, washed a few times with very concentrated KSCN solution, then with hot 80% alcohol, and dried in air. The product is fairly stable and can be stored for months in dry air over  $H_2SO_4$ .

SYNONYM:

Potassium nonachloroditungstate (III).

PROPERTIES:

Formula weight 804.24. Small, dark-green tablets, yellow in transmitted light. Soluble in water, giving a dark-green color; very slightly soluble in alcohol. The aqueous solution oxidizes in air; the crystalline compound also oxidizes but more slowly.

REFERENCES:

O. Collenberg and A. Guthe. Z. anorg. allg. Chem. <u>134</u>, 317 (1924);
O. Collenberg and K. Sandved. Ibid. <u>130</u>, 9 (1923); O. Olson-Collenberg. Ibid. <u>88</u>, 50 (1914); W. Biltz. Ibid. <u>170</u>, 164 (1928);
H. B. Jonassen, A. R. Tarsey, S. Cantor and G. F. Helfrich in: Th. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 139.

#### Hexachlorotripyridineditungstate (III)

## W<sub>2</sub>Cl<sub>6</sub>py<sub>3</sub>

 $\begin{array}{rrrr} K_3 W_2 Cl_9 &+ & 3 \, py &= & 3 \, K Cl &+ & W_2 Cl_6 py_3 \\ 804.3 & & 237.3 & & 223.7 & & 817.9 \end{array}$ 

Ten grams of freshly recrystallized  $K_3W_2Cl_9$  is refluxed in 150 ml. of dry pyridine for 6 hours (N<sub>2</sub> atmosphere). The red solution is filtered to remove the brown precipitate and treated with a large excess (10 volumes) of ether. This precipitates the dark-brown complex salt; it may be recrystallized from pyridine.

PROPERTIES:

Insoluble in water, somewhat soluble in ether and benzene. Diamagnetic. A corresponding aniline complex exists.

REFERENCE:

H. B. Jonassen, S. Cantor and A. R. Tarsey. J. Amer. Chem. Soc. 78, 271 (1956).

#### Potassium Octacyanotungstate (IV)

#### $K_4[W(CN)_8] \cdot 2 H_2O$

Prepared from  $K_3W_2Cl_9$  and KCN.

A solution of 20 g. of  $K_3W_2Cl_9$  in 150 ml. of cold, boiled water is prepared and treated on a water bath with 65 g. of KCN powder; this causes oxidation to W (IV), and the green color of the solution changes to red. The KCN should be added very carefully (shaking) over a period of 5 to 10 minutes.

The solution is now heated about two hours on the water bath, filtered (decolorizing charcoal being added if required) and evaporated until crystals begin to deposit. The first crystal fraction, consisting predominantly of KCl, is filtered off and discarded. The filtrate is diluted to 130 ml. and treated, while still warm, with 20-25 ml. of 95% alcohol. The  $K_4[W(CN)_8] \cdot 2 H_2O$ separates in lustrous, bright-yellow plates on sharp cooling in a freezing mixture. After one hour, 15 additional ml. of alcohol is added. The mixture is suction-filtered after standing for 12 hours in the cold. The product is washed with hot 80% alcohol. Yield: 60-70%, based on tungsten.
For purification, the compound is first precipitated twice from 50% aqueous solution by adding an equal volume of alcohol. It is then dissolved in 16% KCN solution and, after concentrating, allowed to stand at  $0^{\circ}$ C until crystallization is complete. The product is again reprecipitated with water and alcohol to remove traces of KCN.

#### PROPERTIES:

Formula weight 584.5.  $d^{2}\frac{5}{4}$  1.989. Bright-yellow crystalline powder; slow evaporation of a KCN solution affords large, yellowred crystals. Very readily soluble in water (about 13-14 g./10 ml. H<sub>2</sub>O at 18°C). Insoluble in alcohol and ether.

REFERENCE:

O. Olsson-Collenberg. Z. anorg. allg. Chem. <u>88</u>, 50 (1914);
H. Baadsgaard and W. D. Treadwell. Helv. Chim. Acta <u>38</u>, 1669 (1955).

# Potassium Octacyanotungstate (V)

# $K_3[W(CN)_8] \cdot H_2O$

A solution of  $K_4[W(CN)_8] \cdot 2 H_2O$  (11.69 g. = 0.05 moles) in 125 ml. of water (acidified with 2 ml. of conc. HNO<sub>3</sub> and titrated with permanganate to a permanent red color) is prepared. The silver salt is then precipitated by addition of 0.21 g. (0.08 moles) of AgNO<sub>3</sub> dissolved in 50 ml. of water. The solid is washed with dil. HNO<sub>3</sub>, dissolved in the minimum amount of ammonia, and reprecipitated with some dil. HNO<sub>3</sub>. After thorough washing with water, the salt is suspended in 50 ml. of water and converted to the potassium salt by addition of 0.11 g. (0.07 moles) of KCl. The AgCl precipitate is removed by filtration, and the filtrate is treated with alcohol until the  $K_3[W(CN)_8] \cdot H_2O$  precipitates. It is filtered and dried over CaCl<sub>2</sub>. The yield can be as high as 91%, based on the starting tungstate (IV).

PROPERTIES:

Formula weight 527.4. Small lemon-yellow crystals, readily soluble in water.

REFERENCES:

O. Olsson-Collenberg. Z. anorg. allg. Chem. <u>88</u>, 50 (1914);
H. Baadsgard and W. D. Treadwell. Helv. Chim. Acta <u>38</u>, 1669 (1955).

# Uranium

#### U

The preparation of the pure metal is rendered difficult by its great tendency to combine with O, N, C, etc., and to alloy with many metals.

Basically, the following preparative methods are available:

a) Reduction of uranium oxides or halides with suitable metals, such as Na, Mg, Ca (methods I to IV).

b) Electrolysis (method V).

I. In the Jander method,  $UO_2$  is reduced with metallic Ca:

 $UO_2 + 2 Ca = U + 2 CaO$ 270.1 80.2 238.1 112.2

The reactor is an iron crucible about 13 cm. high, 1.6 cm. I.D., with a 0.1-cm. wall. This is charged with the reactant mixture, consisting of 7 g. of  $UO_2$  and 11 g. of Ca turnings (the latter should be as freshly distilled as possible). The crucible cover is then welded on and the reactor heated for one hour at 1000-1100°C (the crucible should be embedded in charcoal powder to protect it from oxidation). The crucible is then completely cooled, opened, and the contents are covered with 90% alcohol saturated with  $NH_4Cl$ . Aqueous  $NH_4Cl$  solution is then added. The material is washed with water and then with alcohol. The product consists of four different fractions:

a) a very finely divided oxide-containing uranium, which can be separated by slurrying;

b) iron metal particles with a small uranium content, which can be removed after drying by means of a magnet;

c) nonmagnetic metal flakes (uranium with a high content of iron), which can be separated on a 140-170 U.S. standard sieve;

d) a very fine gray-brown powder, containing about 97% U and 2-3% Fe, but only very small amounts of O. The yield of this fraction is 66%.

The presence of iron impurity in the product can be avoided by coating the inner surface of the iron crucible with calcium carbonate, but the shrinkage of this lining during drying and its fragility necessitate considerable care in handling. A reactor lined with calcium carbonate is capable of producing material containing 99.9% U, although the yield is appreciably lower than in the above-described procedure.

II. Very pure uranium is obtained by reduction of  $U_3O_8$  with freshly distilled Ca in high vacuum. The mixture is heated above

the melting point of Ca, and the U is obtained as a fine gray powder which can be separated from the by-product CaO by sieving. Any unused Ca (which is present in excess) sublimes out from the product at the reaction temperature. The product analyzes as 99.95% U.

III. Both of the above procedures may be improved by adding to the reactant mixtures a mixture of scrupulously predried and prefused  $CaCl_2$  and  $BaCl_2$ . The salt mixture serves as a flux and at high temperatures dissolves both the CaO and Ca metal. It is further recommended that the product be reduced a second time under the same conditions; this is because the first reduction normally goes to equilibrium and no further.

IV. Many variations of the reaction of uranium chlorides with metallic Na in sealed iron vessels have been described. The products range in purity from 99 to 100% U. The preparative method described below is based on the earlier procedures and attempts to overcome some of their shortcomings; however, it affords a uranium whose x-ray diffraction pattern still clearly shows  $UO_2$  lines. For this reason one must question the assertions of earlier authors who claimed that this procedure gives a completely pure product.

An alumina tube closed at one end (a so-called Tammann crucible), 12 cm. high and 2 cm. I.D., is charged successively with 5 g. of NaCl, 13 g. of uranium chloride (for preparation, see p. 1436 under UCl<sub>4</sub>, method I), and 4.5 g. of Na metal (freshly cut under ether). The materials must be added as rapidly as possible and then ocvered with NaCl up to 1 cm. below the top edge of the crucible. The filled tube is placed inside an only slightly larger iron crucible (3-mm. wall thickness) and the lid of the latter is welded on (see also Part III, Intermetallic Compounds, preparation of alloys by fusion of components). Two thick wires or bars are welded onto the outside of the iron crucible so that it may be suspended in a vertical tubular electric furnace. The reactor is gradually heated to 1150°C (2 hours), held for 15 minutes at this temperature, and then allowed to cool. The crucible is opened, and the contents of the alumina tube are treated with HCl-saturated methanol to remove unreacted Na, then with hot water to dissolve away the NaCl. Yellow Na<sub>2</sub>U<sub>2</sub>O<sub>7</sub> may appear

during this operation, even if the chloride reactant contained only a small percentage of  $UO_2Cl_2$ . However, the  $Na_2U_2O_7$  may be separated completely from the U metal residue by repeated slurrying with water and decantation. The residue is then washed several times with HCl-saturated methanol and with water. Finally, light-gray metallic pellets are obtained.

# V. ELECTROLYTIC PREPARATION

Very pure uranium is obtained by electrolysis of  $KUF_5$  in an NaCl-CaCl<sub>2</sub> melt. A cylindrical graphite crucible serves both as the electrolysis vessel and the anode. It has an I.D. of 6 cm., a height of 15 cm. and walls 1-2 cm. thick. The electrical connection is made with a strip of Ni sheet wrapped around the upper part of the outside wall. The cathode is a strip of Mo sheet, 0.5 mm. thick and 1 cm. wide, which is immersed in the melt so that its lower end is 2.5 cm. above the bottom of the crucible. The latter stands in a suitable refractory vessel, around which a heating wire is wound. The entire apparatus is placed in a large-diameter lead vessel filled with thermal insulation.

A mixture of 250 g. of NaCl and 250 g. of anhydrous CaCl<sub>2</sub> is first fused together, and the melt temperature is adjusted as exactly as possible to 775 °C. When the mixture is thoroughly melted, the cathode is inserted and the current is turned on (30 amp., potential drop between the electrodes about 5 v.). The current density should be 10 amp./in.<sup>2</sup>.

Now, 30 g. of  $KUF_5$  is added in small portions so that it melts as rapidly as possible. After addition of the  $KUF_5$  is completed, a deposit of U begins to form on the cathode, where it appears in the form of a metallic tree; this reaches a thickness of about 2.5 cm. in 45 minutes. The old cathode is then slowly withdrawn from the melt, a new one is introduced, and the electrolysis is continued as before, while fresh salts are added to the melt as needed.

The material adhering to the cathode consists of a gray, spongy mass, which is permeated and surrounded by solidified melt. The melt protects the material from oxidation during cooling. After thorough cooling, the solids are stripped off the cathode and treated with water. Most of the salts dissolve quite easily, while the residue of heavy U powder can be readily and completely freed from traces of  $CaF_2$ , etc., by slurrying with water. The fraction consisting of very fine particles should be separated at the same time, since this material oxidizes very easily. The residue then comprises pure, fairly coarse particles of gray metal. The latter is washed with 5% acetic acid, followed by alcohol and ether; it is dried in vacuum and stored in air-tight vessels. This material is not pyrophoric. However, if the fine metal powder is not removed the dry product may ignite in air under some circumstances.

PROPERTIES:

Light-gray pellets or black powder. M.p. 1689°; d 18,685. May be distilled using an electric furnace. Dissolves in dil. HCl and  $H_2SO_4$  (slowly in cold acids but rapidly in warm ones), evolving  $H_2$ .

**REFERENCES:** 

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# **Uranium Hydride**

#### UH<sub>3</sub>

 $U + 3/2 H_2 = UH_3$ 238.1 33.6 l. 241.1

This procedure is successful only when very pure uranium metal is employed.

Uranium (10 g.) is freed of the adherent oxide layer by brief treatment with dil. nitric acid, washing with water, and drying over  $P_2O_5$ . The metal is then coarsely ground and placed in a porcelain boat, which should be large enough to accommodate the increase in volume which accompanies the hydride formation. Hydrogen is then passed over it at 250°C. [The pretreatment of the H<sub>2</sub> includes passage through a copper column heated to 650-700°C, a drying agent (magnesium perchlorate), and uranium powder heated to 700-750°C.] The reaction is complete after 20-30 minutes; the yield is quantitative.

See also Part II, Section 1, p. 113 f.

PROPERTIES:

Fine, black, pyrophoric powder. Loses  $H_2$  when heated in vacuum above 250 °C, forming a uranium of high chemical activity. Powerful reducing agent; reacts vigorously with water according to: 2 UH<sub>3</sub> + 4 H<sub>2</sub>O = 2 UO<sub>2</sub> + 7 H<sub>2</sub>.

**REFERENCES:** 

F. H. Spedding, A. S. Newton, J. C. Warf, O. Johnson, R. W. Nottorf, J. B. Johns and A. H. Daane. Nucleonics <u>4</u>, 4 (1949).

#### Uranium (III) Chloride

#### UCl<sub>3</sub>

 $UCl_4 + \frac{1}{2}H_2 = UCl_3 + HCl_379.9$  11.2 *l*. 344.4 36.5

I. Completely pure UCl<sub>4</sub> is reduced with  $H_2$  in the same tube in which it is prepared. The reduction proceeds below red heat and then at dull-red heat, until the off-gases are free of HCl. Absolutely pure  $H_2$  must be used. The dull-brown, very hygroscopic UCl<sub>3</sub> adheres strongly to glass. Thus, when the mechanically separated product is dissolved, some brown to brick-red U (IV) silicate is always obtained as a residue.

II. Uranium is reacted with dry HCl at 250-300°C.

PROPERTIES:

Lustrous, dark-red, very hygroscopic needles;  $d_{4}^{25}$  5.440. Very readily soluble in water, giving a purple-red liquid which becomes green within a few seconds as H<sub>2</sub> is evolved and a red precipitate forms. Insoluble in anhydrous alcohol, acetic acid, CCl<sub>4</sub>, CHCl<sub>3</sub>, acetone and pyridine.

REFERENCES:

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  J. F. Suttle in: Th. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 145.
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# Uranium (IV) Chloride

# UCl<sub>4</sub>

I. The preparation of completely pure  $UCl_4$  (or  $UCl_5$ ) is difficult since under normal conditions there exists an equilibrium: 2  $UCl_4 + Cl_2 = 2 UCl_5$ . However, for many purposes, especially for the subsequent reduction to metallic U, any mixture of the two uranium chlorides is adequate. Such a mixture is prepared rapidly and conveniently by the following method.

A quartz tube 40 cm. long and 2 cm. I.D., both ends of which carry ground joints, is charged with a mixture of 20 g. of UO<sub>2</sub> and 7 g. of carbon black, distributed along the entire length of the tube. One end of the reactor tube is connected to a wash bottle containing  $H_2SO_4$ ; the latter, in turn, is connected to a U tube filled with  $P_2O_5$ , which is attached to a  $Cl_2$  cylinder. The other end of the reactor is connected to two ground-joint Erlenmeyer flasks in series (do not use round-bottom flasks; these are not as well suited for the precipitation operation), which in turn are attached to a wash bottle with H<sub>2</sub>SO<sub>4</sub> (the latter is connected backward and serves as a trap for atmospheric moisture). All joints are lubricated with vitreous phosphoric acid. A very fast stream of  $Cl_2$  is passed through this apparatus. When all the air is displaced, the reactor is heated with two rosette burners, beginning at the Cl<sub>2</sub> inlet. Within a short time the uranium chloride begins to sublime into the cooler part of the tube, largely as a brown vapor. By shifting the burners to the next zone of the tube as the reaction is completed in the preceding one, it is possible to sublime most of the uranium chloride into the two Erlenmeyer flasks. The yield is almost quantitative. It takes one hour to obtain about 13 g. of uranium chloride from 20 g. of  $UO_2$ .

Chlorination of  $U_3O_8$  under identical conditions, sometimes recommended in the literature, gives chlorides greatly contaminated with  $UO_3Cl_2$ .

II. UCl<sub>4</sub> may be obtained by heating a mixture of 20 g. of UO<sub>2</sub> and about 6 g. of sugar-derived charcoal, covered with some additional sugar charcoal powder. The reactants are in an unglazed boat and a stream of Cl<sub>2</sub> is passed over the latter (the air is first displaced by evacuation and several purgings with Cl<sub>2</sub>). The reaction begins at 450°C and is completed at 600-700°C. The UCl<sub>4</sub> is driven into a spherical receiver sealed onto the reactor tube to decompose the by-product UCl<sub>5</sub>. The material must be redistilled in a CO<sub>2</sub> stream into a second receiver sealed onto the first. III.

 $UO_2 + 2 SOCl_2 = UCl_4 + 2 SO_2$ 270.1 237.9 379.9 128.1

A bomb tube of  $\sim 1.5$  cm. I.D. is charged with 5.6 g. of pure UO<sub>2</sub> and 10 ml. of SOCl<sub>2</sub> (freshly distilled in vacuum); these are then thoroughly mixed. The sealed tube is heated for 7 days at 200°C; the time required for completion of the reaction may be somewhat reduced if the tube is briefly cooled at regular intervals and the SO<sub>2</sub> formed is permitted to escape.

The product consists of  $UCl_4$ , partially dissolved in  $SOCl_2$  and partially present as green crystals. The lower part of the bomb tube is cut off, the entire contents rinsed rapidly with some  $SOCl_2$  into a 100-ml. ground-joint flask and distilled under reduced pressure (e.g., 140 mm.) in a stream of dry  $N_2$  (or  $CO_2$ ) to remove the  $SOCl_2$ . Finally, the brown adduct of  $SOCl_2$  and  $UCl_4$  is decomposed by heating at about 150°C until a pure green residue of  $UCl_4$  is obtained (half an hour is required). Yield: about 7.5 g. (95%).

IV. Analogous to Hönigschmid's method for  $UBr_4$  (see p. 1440 f.). The product has the composition indicated by the formula. The procedure is somewhat tedious because of numerous precautionary measures necessary. Useful only for small quantities.

V. Chlorination of UCl<sub>3</sub> at 250°C.

PROPERTIES:

Formula weight 379.90. Light-green needles or dark-green octahedra, which sublime at red heat as a red vapor. M.p. 567°, b.p.  $618^{\circ}$ ;  $d^{25}_{4}$  4.73-4.97. The aqueous solution gives a strongly acidic reaction because of hydrolysis. Soluble in ethyl acetate and benzoate; insoluble in ether, chloroform and benzene.

REFERENCES:

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- V. E. Staritzky. Analyt. Chem. <u>28</u>, 1056 (1956). See also J. A. Hermann and J. F. Suttle in: Th. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 143.

# Uranium (V) Chloride

#### UCl<sub>5</sub>

$$\begin{array}{rcl} UCl_4 &+ & \frac{1}{2} Cl_2 &= & UCl_5 \\ 379.9 & & 11.0 \ l. & & 415.4 \end{array}$$

A very clean hard-glass tube a-h, bent as shown in Fig. 324, is charged at location b with a mixture of UCl<sub>4</sub> and finely divided wood charcoal, and attached immediately to a Cl<sub>2</sub>-generating apparatus via joint a. The reactants and the glass tube are carefully dried by heating in the Cl<sub>2</sub> stream, which is predried over P<sub>2</sub>O<sub>5</sub>. When stronger heating is applied at b, a mixture of UCl<sub>4</sub> and UCl<sub>5</sub> distills toward d. To completely convert this chloride mixture into UCl<sub>5</sub>, a sufficient quantity of Cl<sub>2</sub> is first frozen in f by cooling the latter; the tube is then sealed off at c and (after evacuation via joint h) at g. The solid Cl<sub>2</sub> at f is brought to 0°C. This results in a vapor pressure of about 3.6 atm. Then heating the tip of tube dcauses sublimation of pure UCl<sub>5</sub> from the heated zone; it forms a dark-brown deposit at e.



Fig. 324. Preparation of uranium (V) chloride.

Alternate method: Chlorination of  $U_3O_8$  with  $CCl_4$  in a sealed tube at 250°C (Michael and Murphy).

#### PROPERTIES:

Deep-brown, crystalline, very hygroscopic sublimate. Dissociates slowly, even at room temperature, to UCl<sub>4</sub> and Cl<sub>2</sub> (equilibrium partial pressure of Cl<sub>2</sub> at 20°C is at least  $10^{-2}$  mm.); must therefore be stored in sealed vessels filled with Cl<sub>2</sub>. Soluble in water with fizzing and evolution of HCl; soluble in absolute alcohol and acetone; the best solvents are ethyl acetate and benzonitrile.

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 A. Michael and A. Murphy. J. Amer. Chem. Soc. <u>44</u>, 365 (1910).

# Uranyl Chloride $UO_2Cl_2$ $UO_2SO_4 \cdot 3 H_2O + BaCl_2 = UO_2Cl_2 + BaSO_4 + 3 H_2O$ $420.2 \qquad 208.3 \qquad 341.0$ $UO_3 + 2 HCl = UO_2Cl_2 + H_2O$ $286.1 \qquad 72.9 \qquad 341.0$ $U_3O_8 + 6 HCl + H_2O_2 = 3 UO_2Cl_2 + 4 H_2O$ $842.2 \qquad 218.8 \qquad 34.0 \qquad 1023.0$

An aqueous solution of uranyl chloride is prepared by (a) dropwise addition of a BaCl<sub>2</sub> solution to a conc. solution of  $UO_2SO_4$ .  $3 H_2O$  [for preparation, see p. 1447 under uranium (IV) sulfate] until all of the  $SO_4^{3-}$  ion is precipitated (no excess of BaCl<sub>2</sub>), followed by filtration; or (b) by thoroughly boiling  $UO_3$  with water to give a yellow powder of  $H_2UO_4$ , which is then dissolved in dil. HCl; or (c) by slurrying  $U_3O_8$  in conc. HCl, followed by dropwise addition of  $30\% H_2O_2$ ; the  $U_3O_8$  is thus dissolved as  $UO_2Cl_2$ , although the reaction is slow.

Careful evaporation of any of the  $UO_2Cl_2$  solutions on a water bath and then in a vacuum desiccator over conc.  $H_2SO_4$  affords a crystalline mass of composition  $UO_2Cl_2 \cdot H_2O$ . To obtain  $UO_2Cl_2 \cdot$  $3 H_2O$ , a small portion of the residue is heated with some conc. HCl, the resulting solution is allowed to evaporate in a desiccator, and the small crystals which separate are added to a conc. solution of the main body of the monohydrate, whereupon they grow into large, uniform prismatic crystals.

The  $UO_2Cl_2 \cdot H_2O$  and  $UO_2Cl_2 \cdot 3 H_2O$  can be dehydrated without decomposition by the following method. The uranyl chloride is first dried over  $P_2O_5$ , placed in flat porcelain boats, and slowly heated in a dry  $HCl/Cl_2$  stream to about 450°C over a period of 4-5 hours. Under these conditions no decomposition to the oxide takes place and only the water of crystallization is removed. If any  $UO_2(OH)Cl$  is present, it is converted to  $UO_2Cl_2$  by reaction with the HCl, liberating water.

Alternate methods: a) repeated evaporation of uranyl nitrate or acetate with conc. HCl yields  $UO_2Cl_2$  solutions, which are crystallized in a vacuum desiccator over KOH. The amount of water of crystallization present in the product depends on the duration of the drying period.

b) Oxidation of  $UCl_4$  with  $O_2$  (Leary and Suttle).

PROPERTIES:

UO2Cl3

Golden-yellow when completely anhydrous; hydrated  $UO_2Cl_2ex$ -hibits a greenish luster.

 $UO_2Cl_2 \cdot 3 H_2O$ 

Formula weight 395.03. Yellow-green, fluorescent, obliqueangled prisms, deliquescent in air; extremely soluble in water, alcohol and ether. At 18°C, 7.35 wt. parts of  $UO_2Cl_2 \cdot 3 H_2O$  dissolve in 1 part of  $H_2O$ ; the saturated solution is very viscous.

REFERENCES:

F. Mylius and R. Dietz. Ber. dtsch. chem. Ges. <u>34</u>, 2774 (1901);
W. Oechsner de Coninck. Comptes Rendus Hebd. Séances Acad. Sci. <u>148</u>, 1769 (1909); L. Ochs and F. Strassmann. Z. Naturforsch. <u>7</u> b, 637 (1952); H. A. Leary and J. F. Suttle in: Th. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 148.

# Uranium (IV) Bromide

#### UBr<sub>4</sub>

Prepared by brominating a mixture of  $UO_2$  and charcoal.

The apparatus used by Hönigschmid (Fig. 325) consists essentially of two parts: the glass section A and the quartz tube B with receiver C. Section A serves to hold the weighing tube and its stopper, and is attached to the quartz tube B by means of a large flange joint. Section B has a sacklike protuberance on one side, of the same I.D. as the quartz tube itself.

System D, comprising 3 quartz tubes connected by ground joints, is inserted into tube B. The side view of this system is shown separately in the figure, and is also reproduced in the main drawing. Tube a-d, constricted in the middle, carries a quartz boat containing the oxide-charcoal mixture. Tube b, which will be stored later in the weighing tube mentioned above, serves as a receiver for the pure, fused UBr<sub>4</sub>. Tube c leads the uncondensed bromide vapors into the receiver.

The quartz boat is charged with a mixture of 1 part of sugarderived charcoal and 4 parts of uranium oxide, intimately ground together in an agate mortar. The boat is then inserted into tube a. The flanged joint and all stopcocks which will come in contact with bromine vapor are greased with sirupy metaphosphoric acid. The flanged connection is held together with strong metal springs. The apparatus is heated with small electric tube furnaces which can be shifted along the length of the quartz tube as far as the protuberance.

The apparatus is first filled with  $N_2$ , and the quartz tube is heated along its entire length (beginning at the protuberance) in order to dry the material. Then a Br<sub>2</sub>-saturated stream of  $N_2$  is

introduced and the furnaces are shifted in such a manner that the entire system of tubes can be heated as far as part d. The temperature is raised to yellow heat; the UBr<sub>4</sub> begins to form and condenses in d. Tubes b and c remain completely free of material and only a small amount of UBr<sub>4</sub> collects in receiver C. About 5 g. of UBr<sub>4</sub> forms in one hour. The furnaces are now shifted in such a way that the preweighed tube b remains cold, yet the sublimate in d is heated to yellow heat. The second sublimation (from d into b) is carried out either in bromine vapor or in pure N<sub>2</sub>.



Fig. 325. Preparation of uranium (IV) bromide.

The melting of the  $UBr_4$  should be accomplished without a loss, if possible, and without too long an exposure to high temperature. Thus, after completion of the sublimation, the entire system of tubes is pushed toward the protuberance by means of a quartz rod inserted through receiver C, while all furnaces are still at their maximum temperature. In this way the small tube b with the  $UBr_4$ is shifted into the hottest part of the apparatus. The sublimate fuses in a few moments; the furnaces are now shut down and removed at once. Unnecessary overheating of the  $UBr_4$  is thus avoided.

The apparatus is allowed to cool in a stream of  $N_2$ , and are then filled with dry air. In disassembling the tube system, receiver *C* is removed first and the individual joints are loosened with a long glass rod provided with a small hook, while a steady, fast stream of air is passed through. Tube *a* is pushed up to the protuberance and is allowed to glide into the latter by gently turning the whole apparatus. Then tube *b* with the fused sublimate is pushed into the previously prepared weighing tube, which is then closed in the usual manner. The length of the run from the beginning of heating to the disassembly of the apparatus is 3 hours or less.

PROPERTIES:

Formula weight 557.73.  $d^{2}_{4}$  4.838. Lustrous, brown to black leaflets, sublimable in a Br<sub>2</sub>-N<sub>2</sub> stream; in N<sub>2</sub> alone, dissociates partially to UBr<sub>3</sub> and Br<sub>2</sub>. Dissolves in H<sub>2</sub>O with fizzing and formation of a green liquid.

REFERENCES:

O. Hönigschmid. Monatsh. Chem. <u>36</u>, 59 (1915); O. Hönigschmid and F. Wittner. Z. anorg. allg. Chem. <u>226</u>, 296 (1936).

# Uranium (IV) Oxide

#### UO2

I.  $U_3O_8 + 2H_2 = 3UO_2 + 2H_2O_842.2 \quad 44.8 \, l. \quad 810.2 \quad 36.0$ 

The starting  $U_3O_8$  is prepared by heating pure uranyl nitrate, oxalate or peroxide (or ammonium diuranate) to 700-800 °C; it is then reduced with  $H_2$  at 900 °C and allowed to cool in the stream of  $H_2$ .

II.  $UO_2C_2O_4 \cdot 3H_2O = UO_2 + 2CO_2 + 3H_2O$ 412.1 270.1

Precipitation of a hot conc. solution of uranylnitrate with oxalic acid yields a yellow powder of  $UO_2C_2O_4 \cdot 3 H_2O$ ; this is converted to black, very fine, pyrophoric  $UO_2$  powder in a stream of  $H_2$  even below red heat.

REFERENCES:

Formula weight 270.1. Brown powder. M.p. 2176° under N<sub>2</sub>;  $d_{\frac{25}{4}}^{2}$  10.8. Crystal structure: C 1 (fluorite) type.

REFERENCES:

I. W. Biltz and H. Müller. Z. anorg. allg. Chem. <u>163</u>, 261 (1927). II. W. Jander. Ibid. <u>138</u>, 321 (1924).

# Uranium (VI) Oxide

#### UO<sub>3</sub>

Pure UO<sub>3</sub> is difficult to prepare because the thermal cleavage of uranyl compounds does not free the product of traces of volatile components, while at high temperatures dissociation into  $U_3O_8$  and  $O_2$  becomes objectionable. To circumvent these drawbacks it is desirable to use  $O_2$  at a pressure above atmospheric.

I.  $UO_4 \cdot 2 H_2O = UO_3 + \frac{1}{2}O_2 + 2 H_2O$ 338.1 286.1

A weighing tube is charged with 5-10 g. of the dry peroxide and placed (unstoppered) in an electric crucible furnace preheated to  $350^{\circ}$ C. A fast stream of  $O_2$  is admitted through the opening in the

furnace lid. The temperature is initially held for 3-5 hours at 350°C and then for one half to one hour at 400°C. The weighing tube is then stoppered and allowed to cool in a desiccator.

II. Uranyl nitrate is heated in  $O_2$  to 500°C; however, the product still contains traces of water.

III.  $U_3O_8 + \frac{1}{2}O_2 = 3 UO_3$ 842.2 858.2

Figure 326 shows the apparatus in which rather large quantities of  $U_3O_8$  can be converted to  $UO_3$  at an oxygen pressure higher than atmospheric. The reactor r consists of an Inconel (a Cr-Ni alloy) tube which is screwed into a brass flange plate. This plate also carries a seal seat groove with a neoprene gasket. The upper, blind flange plate, also made of brass, carries the seal tongue and is drilled for a welded-on brass cross. The latter is connected to two high-pressure diaphragm valves and a pressure gage. A removable quartz insert q facilitates replacement of reactants. Manometer m is arranged in such a way that it also serves as a pressure-relief valve. The graduated tube g is for the liquid oxygen; it must contain sufficient oxygen to generate the desired pressure in the reactor.

Pure  $U_3O_8$  (or  $UO_3$  prepared as in method I) is placed in the quartz insert tube g and the reactor assembled to the flanges. The entire system is purged with  $O_2$ , beginning at stopcock  $h_1$ ; the  $O_2$ may be discharged at the pressure gage. Then stopcock  $h_1$  is closed and the apparatus is evacuated by opening stopcocks  $h_2$  and  $h_3$ ; in this operation, value  $v_4$  is open and  $v_5$  closed. The U<sub>3</sub>O<sub>8</sub> is completely dried by heating reactor r for one hour at 850°C while maintaining the vacuum. A McLeod gage is used to ascertain when the apparatus is completely evacuated; the vacuum connection at stopcock  $h_2$  is then closed, as are stopcock  $h_3$  and valve  $v_4$ . Trap f is filled with liquid  $O_2$  by immersing it in liquid  $N_2$  and opening  $h_1$ . The amount of O<sub>2</sub> condensed in the trap should exceed by 10% that required to bring the pressure in the reactor to 27 atm. gage. This O<sub>2</sub> quantity can be estimated more exactly after the apparatus has been used once. When sufficient  $O_2$  is condensed in the trap, stopcock  $h_1$  is closed. Now the graduated tube g is immersed in a Dewar flask filled with liquid  $N_2$  and stopcock  $h_3$  is opened. The exact quantity of O<sub>2</sub> which is needed to attain the required gage pressure is measured into g and stopcock  $h_3$  is then closed. The reactor is cooled in liquid N<sub>2</sub> and value  $v_4$  is opened. Tube g is then immersed in liquid O<sub>2</sub> in order to establish an inside pressure of about 1 atm.; this causes the O<sub>2</sub> to distill over into the reactor. Next, value  $v_4$  is closed and the reactor r is brought very gradually to room temperature in order to avoid scattering of the  $U_3O_8$  by

the evaporating  $O_2$ . The best procedure for doing this is to replace the liquid- $N_2$ -filled Dewar flask with an empty one. The reactor tube is placed in an electric furnace and heated for 40 hours at 600-700 °C. If too much  $O_2$  is condensed in the reactor, the excess can be discharged through valve  $v_5$ . The liquid  $O_2$  remaining in trap f can be removed by vacuum, or allowed to escape (slowly) through the manometer while the trap is in a Dewar flask containing gradually evaporating liquid  $N_2$ .

According to x-ray diffraction data, this procedure yields pure  $UO_3$ .



Fig. 326. Preparation of uranium (VI) oxide. r Inconel reactor; q quartz insert tube; ggraduated condensation trap; f condensation trap; m manometer and pressure-relief valve; v valves.

PROPERTIES:

Bright orange-yellow, very hygroscopic, amorphous powder or hexagonal crystals;  $d_{4}^{25}$  7.368. Soluble in mineral acids, forming uranyl salts. In water, it hydrates in 24 hours at room temperature to give  $UO_3 \cdot H_2O_3$ .

A red, hexagonal modification, which is less stable, forms at 450-500 °C; its crystal structure resembles that of  $U_3O_8$ .

**REFERENCES:** 

- I. W. Biltz and H. Müller. Z. anorg. allg. Chem. <u>163</u>, 258 (1927).
- II. G. F. Hüttig and E. v. Schroeder. Ibid. <u>121</u>, 250 (1922); S. S. Lu. Sci. Technol. China <u>1</u>, 12 (1948), abstr. in Chem. Zentr. <u>1949</u>, II, 951.
- III. J. Sheft, S. Fried and N. Davidson. J. Amer. Chem. Soc. <u>72</u>, 2172 (1950).

# Alkali Uranates (VI)

# $Li_2UO_4$ , $Na_2UO_4$ , $K_2UO_4$

 $UO_3 + Li_2CO_3(Na_2CO_3, K_2CO_3) = Li_2UO_4(Na_2UO_4, K_2UO_4) + CO_2$ 286.1 73.9 (106.0, 138.2) 316.0 (348.1, 380.3)

Alkali carbonate and UO<sub>3</sub> (1:1 mole ratio) are intimately ground, placed in a large-diameter crucible, and gradually heated in an electric furnace to 800 °C while  $O_2$  is slowly passed through. The process is interrupted several times to regrind the reactants. The end of the reaction is recognized by the failure of a sample to evolve  $CO_2$  on dissolution in dil. hydrochloric acid (2-3 days required to reach this point).

PROPERTIES:

Light orange when finely divided. Aqueous slurries show a distinctly alkaline reaction within a few minutes. Soluble in dilute hydrochloric acid and sulfuric acid, as well as in 2 N acetic acid (except for potassium uranate).

REFERENCE:

W. Rüdorff and H. Leutner. Z. anorg. allg. Chem. 292, 193 (1957).

#### Alkali Uranates (V)

#### LiUO<sub>3</sub>, NaUO<sub>3</sub>

 $Li_2UO_4(Na_2UO_4) + UO_2 = 2 LiUO_3(NaUO_3)$ 315.9 (348.0) 270.1 586.0 (618.1)

A mixture of alkali uranate (VI) and  $UO_2$  (1:1 mole ratio) is heated in evacuated, sealed quartz ampoules at 650-750°C. The ampoules are opened at intervals of 10-20 hours, the reaction product is reground to a fine powder, and the heating is resumed using a new ampoule. The reaction is complete in 75-100 hours; the lithium compound forms in a somewhat shorter time and at a somewhat lower temperature.

PROPERTIES:

Formula weights:  $LiUO_3$  293.0;  $NaUO_3$  309.1.  $LiUO_3$  is dark violet,  $NaUO_3$  brown-violet. Both are very stable. Much more resistant to acids than the corresponding uranates (VI). Dilute hydrochloric and sulfuric acids have no effect in the cold. Dissolve in dil. nitric acid. REFERENCE:

W. Rüdorff and H. Leutner. Z. anorg. allg. Chem. 292, 193 (1957).

# Uranium Peroxide

# $UO_4 \cdot 2 H_2O$

Precipitates from uranyl nitrate solutions on addition of H<sub>2</sub>O<sub>2</sub>.

I. A boiling 10% solution of uranyl nitrate is treated dropwise with 30% H<sub>2</sub>O<sub>2</sub>. The resulting amorphous, white precipitate is filtered on the finest filter possible (membrane or Millipore filter) and washed thoroughly with boiling water. The peroxide, which is a bright sulfur-yellow after filtration, is first dried in the air on a clay plate, then at 100°C to constant weight, and stored in vacuum over P<sub>2</sub>O<sub>5</sub>.

II. Reaction of  $(NH_4)_2[UO_2(C_2O_4)_2] \cdot 3 H_2O$  with  $H_2O_2$  yields crystalline, nonhygroscopic  $UO_4 \cdot 3 H_2O$ , which is converted to the dihydrate by storing in vacuum over  $P_2O_5$ .

III. A readily filterable peroxide hydrate is obtained from the reaction of 50 ml. of 30% H<sub>2</sub>O<sub>2</sub> with 3 g. of UO<sub>3</sub> (half a day at room temperature).

PROPERTIES:

Formula weight of  $UO_4 \cdot 2 H_2O$ : 338.10. Yellowish-white, amorphous powder or fine needles.

**REFERENCES**:

- I and II. A. Rosenheim and H. Daehr. Z. anorg. allg. Chem. <u>181</u>, 178, 180 (1929).
- III. A. Sieverts and E. Müller. Ibid. <u>173</u>, 299 (1928).

# Uranium (IV) Sulfide

US<sub>2</sub>

 $UCl_4 + 2 H_2S = US_2 + 4 HCl$ 379.9 44.8 l. 302.2

I. The best starting material for the preparation of  $US_2$  is  $Na_2UCl_6$ ; this material is preferable to  $UCl_4$  because its volatility is lower.

The starting mixture is prepared during the synthesis of UCl<sub>4</sub> (see p. 1436, method II); thus, the sealed-on round receiving flask is precharged with 10 g. of ignited NaCl. Fusion of the UCl<sub>4</sub> with NaCl yields a green cake. The Cl<sub>2</sub> is removed by evacuation of the reactor tube. Then a stream of dry  $H_2S$  (either generated from the liquefied material or made by passing pure, dry  $H_2$  over a boat containing molten S) is passed over the Na<sub>2</sub>UCl<sub>6</sub> while heating the latter to 600-700°C; the reaction is continued for 4-5 hours until the off-gases are free of HCl. The US<sub>2</sub> is allowed to cool under  $H_2S$  and washed briefly with deaerated ice water, then with alcohol and ether, and dried in vacuum at 140°C.

II. Prepared from  $U_3O_8$  and  $H_2S$  at 1150 °C (electric furnace); depending on the reaction conditions, either  $\alpha$  - or  $\beta$  -US<sub>2</sub> is formed.

#### PROPERTIES:

Black leaflets with a metallic luster, altered only by prolonged standing in air;  $d_{4}^{25}$  7.96.

#### REFERENCES:

- I. E. F. Strotzer, O. Schneider and W. Biltz. Z. anorg. allg. Chem. 243, 307 (1940); A. Colani. Ann. Chim. Phys. (8) 12, 80 (1907); R. Flatt and W. Hess. Helv. Chim. Acta 21, 526 (1938).
- II. M. Picon and J. Flahaut. Comptes Rendus Hebd. Séances Acad. Sci. 237, 808 (1953).

# Uranium (IV) Sulfate

#### U(SO<sub>4</sub>)<sub>2</sub> · 8 H<sub>2</sub>O or 4 H<sub>2</sub>O

Prepared by cathodic reduction of UO<sub>2</sub>SO<sub>4</sub>:

 $\begin{array}{rrrr} UO_2SO_4 &+& H_2SO_4 &+& 2\,e \longrightarrow U(SO_4)_2\\ (\cdot 3\,H_2O) &&& (\cdot 8\,H_2O)\\ 420.2 && 98.1 && 574.3 \end{array}$ 

The electrolyte consists of an approximately saturated solution of  $UO_2(SO_4) \cdot 3 H_2O$  (1 mole) in about twice the stoichiometric quantity (2 moles) of  $H_2SO_4$ . The  $UO_2SO_4 \cdot 3 H_2O$  is prepared from  $UO_2(NO_3)_2 \cdot 6 H_2O$  by evaporating to dryness with  $H_2SO_4$  and concentrating an aqueous solution of the residue to a sirupy consistency, whereupon the  $UO_2SO_4 \cdot 3 H_2O$  slowly crystallizes. The  $UO_2SO_4-H_2SO_4$  solution is placed in a cooled glass cylinder which houses the cathode space. A clay cell contains the anode. The best results are obtained with an Hg cathode and a carbon rod anode. The Hg layer is placed on the bottom of the cathode vessel; the electrical connection is made with a copper wire sealed into a glass tube. The reaction proceeds quite rapidly at 3-5 amp. Any material which separates during the electrolysis is redissolved by addition of some water. At the end the cathode liquid becomes dark green with a steel-blue to black-violet fluorescence. If the electrolysis continues beyond the tetravalent state, the reddish brown color (in transmitted light) of trivalent uranium becomes apparent; however, this compound is very unstable and is quickly reoxidized in air to tetravalent uranium.

The concentrated acidic  $U(SO_4)_2$  solution thus obtained is very stable and may be kept for weeks, in contrast to the very dilute solution.

Concentration of the solution in vacuum over  $H_2SO_4$  yields  $U(SO_4)_2 \cdot 8 H_2O$  as large, dark-green crystals; alternately, the product may be obtained by evaporation in air below 75°C. If the product is precipitated with alcohol in the cold, it is a light-green, fine crystalline powder.

The tetrahydrate  $U(SO_4)_2 \cdot 4H_2O$  is prepared by dropwise addition of conc.  $H_2SO_4$  to the reduced solution (high-speed stirring); the  $H_2SO_4$  is added until no further precipitation occurs. During this operation the temperature rises to 40-50 °C.

The salt is washed with alcohol and ether and dried on a clay plate.

PROPERTIES:

 $U(SO_4)_2 \cdot 8 H_4O$ : Formula weight 574.32. Dark-green monoclinic prismatic crystals. Hydrolyzes on solution in water, precipitating the basic sulfate  $UOSO_4 \cdot 2 H_2O$ . Solubility (20°C): 8.78 g./100 g. of solution in 0.1 N H<sub>2</sub>SO<sub>4</sub>.

 $U(SO_4)_2 \cdot 4 H_2O$ : Formula weight 502.26. Whitish-green precipitate composed of needles arranged in a starlike form. Soluble in water with separation of the basic sulfate; soluble in dilute acids.

REFERENCE:

R. J. Meyer and H. Nachod. Liebigs Ann. <u>440</u>, 186 (1924).

#### Ammonium Uranyl Carbonate

#### (NH<sub>4</sub>)<sub>4</sub>[UO<sub>2</sub>(CO<sub>3</sub>)<sub>3</sub>]

1044.5

The  $(NH_4)_2U_2O_7$  is precipitated from an aqueous solution of 10 g. of  $UO_2(NO_3)_2 \cdot 6 H_2O$  by addition of conc. ammonia. The fine yellow powder is suction-filtered, washed with water, and stirred with an excess of conc.  $(NH_4)_2CO_3$  solution for about 10 minutes (the flask is on a 70°C water bath). The clear supernatant liquid is decanted and allowed to stand overnight. Yellow crystals precipitate; these are filtered with suction and dried in air. The residue of undissolved  $(NH_4)_2U_2O_7$  is treated several times with the mother liquor at 70°C, as described above, until crystals no longer form on cooling. Yield: 5-8 g.

PROPERTIES:

624.2

Formula weight 522.26. Well-formed, transparent yellow crystals, monoclinic prismatic; d 2.773. Sparingly soluble in water; insoluble in alcohol and ether; may be recrystallized without decomposition from aqueous  $(NH_4)_2CO_3$ .

**REFERENCE:** 

Ebelmen. Liebigs Ann. 43, 302 (1842).

# Uranium (IV) Oxalate

#### $U(C_2O_4)_2 \cdot 6 H_2O$

 $\begin{array}{rl} UO_{2}(CH_{3}COO)_{2} + 4 \ HCl + \ Na_{2}S_{2}O_{4} = UCl_{4} + 2 \ NaHSO_{3} + 2 \ CH_{3}COOH \\ & (2 \ H_{2}O) \\ & 424.2 & 145.9 & 210.1 \end{array}$ 

 $UCl_4 + 2H_2C_2O_4 + 6H_2O = U(C_2O_4)_2 \cdot 6H_2O + 4HCl$ 180.1 522.2

Five grams (0.012 moles) of  $UO_2(CH_3COO)_2 \cdot 2 H_2O$  powder is dissolved in 100 ml. of dilute HCl (1:10 in water) preheated to 80°C. While stirring, 5 g. (0.024 moles) of  $Na_2S_2O_4 \cdot 2 H_2O$  powder is

added in small portions. The initial precipitate is brown, but rapidly changes to whitish-green. Then 5 ml. of conc. HCl is added and the mixture is heated for about 10 minutes on the water bath (until solution is complete). The dark-green solution of uranium (IV) salt is usually somewhat cloudy because of a haze of fine sulfur. It is filtered in the absence of air and treated while still warm (appr.  $60^{\circ}$ C) with a saturated oxalic acid solution; the latter is added slowly (good stirring). A heavy, solid gray precipitate forms at once. It settles in a few minutes and, after standing for one half hour, exhibits the dark-green color of uranium (IV) oxalate. It is washed 5 times with 100-ml. portions of water. Sulfite and oxalate should be removed completely by this operation. Since uranium (IV) oxalate is completely stable in air, it may be air dried. The yield is almost quantitative (5.7 g.).

PROPERTIES:

Dark-green microcrystals, stable in air [in contrast to solutions of uranium (IV) salts]. May be recrystallized from warm conc. hydrochloric acid. Only slightly soluble in water and dil. acids. Loses 5 moles of  $H_2O$  at 110°C, but the sixth mole only at about 200°C.

REFERENCES:

V. Kohlschütter and H. Rossi. Ber. dtsch. chem. Ges. <u>34</u>, 1473, 3630 (1901); E. Marchi in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 166.

# Potassium Tetraoxalatouranate (IV)

#### $K_4[U(C_2O_4)_4] \cdot 5H_2O$

 $\begin{array}{rrrr} U(C_2O_4)_2 \cdot 6 \ H_2O \ + \ 2 \ K_2C_2O_4 \cdot H_2O \ = \ K_4[U(C_2O_4)_4] \cdot 5 \ H_2O \ + \ 3 \ H_2O \\ 522.2 \ & 368.5 \ & 836.6 \end{array}$ 

A slurry of 6 g. (0.014 moles) of  $U(C_2O_4) \cdot 6 H_2O$  in 50 ml. of water is treated in the absence of air with a solution of 5 g. (0.027 moles) of  $K_2C_2O_4 \cdot H_2O$  in 20 ml. of water and allowed to stand on a steam bath for one hour. It is then filtered and the dark-green filtrate is treated dropwise with 200 ml. of absolute alcohol (good stirring). Small light-green crystals precipitate. These are filtered off, washed with absolute alcohol, then with ether, and dried over  $P_2O_5$ .

PROPERTIES:

Readily soluble in water (21.7 g. per 100 g. of water at 17°C), but only very slightly soluble in water-alcohol mixtures. Converted to the monohydrate by heating for a few hours at 200°C.

**REFERENCES:** 

V. Kohlschütter. Ber. dtsch. chem. Ges. <u>34</u>, 1472, 2619 (1901);
E. Marchi in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 169.

#### Uranium (V) Ethoxide

#### $U(OC_2H_5)_5$

 $UCl_{4} + 4 \operatorname{NaOC}_{2}H_{5} = U(OC_{2}H_{5})_{4} + 4 \operatorname{NaCl}$  $U(OC_{2}H_{5})_{4} + \frac{1}{2} \operatorname{Br}_{2} = U(OC_{2}H_{5})_{4} \operatorname{Br}$  $U(OC_{2}H_{5})_{4}\operatorname{Br} + \operatorname{NaOC}_{2}H_{5} = U(OC_{2}H_{5})_{5} + \operatorname{NaBr}$ 

A sodium ethoxide solution is prepared from 800 ml. of absolute ethanol and 46 g. (2 g.-atoms) of Na (use a 1-liter, three-neck flask). Toward the end of the reaction, refluxing and good stirring are needed. The solution is cooled to room temperature. Then, while stirring rapidly, 190 g. (0.5 moles) of fine UCl<sub>4</sub> powder is added in portions of about 20 g. (5-minute intervals). The contents of the flask are protected at all times against atmospheric moisture by means of a CaCl<sub>2</sub> tube. The heat of reaction causes the alcohol to boil, and the flask is therefore set in cold water. When all the UCl<sub>4</sub> has been added and the reaction subsides, the flask contents are refluxed on a steam bath for two hours (stirring). They are then cooled to room temperature and a solution of 40 g. (0.5 g.-atoms) of bromine in 20 ml. of dry benzene is added dropwise (rapid stirring) over a period of 15 minutes. The color changes from light green to brown, then gray and, toward the end of the addition, dark green.

While continuing the stirring, a sodium ethoxide solution prepared from 11.5 g. (0.5 g.-atoms) of Na in 200 ml. of absolute alcohol is rapidly added, causing the color to turn brown. The mixture is then distilled under anhydrous conditions to remove the alcohol. To achieve this, about one third of the material is introduced as rapidly as possible into a 500-ml. Claisen flask and the alcohol is distilled off on an oil bath (good stirring). Then the second fraction is added, etc. When the mass becomes solid, the stirrer is removed; the flask is closed off with a stopper and gradually heated to  $150^{\circ}$ C at 2-3 mm. The completely dry residue, which contains NaCl and U(OC<sub>2</sub>H<sub>5</sub>)<sub>5</sub>, is now heated further on an oil bath at a vacuum of 0.001-0.004 mm. The uranium (V) ethoxide distills at a bath temperature of about  $180-240^{\circ}$ C. The yield may be as high as 217 g. (94%). SYNONYM:

Uranium (V) ethylate.

PROPERTIES:

Dark-brown liquid. B.p. about  $123^{\circ}$ C at 0.001 mm.; d<sup>25</sup> 1.711. Considerably higher thermal stability than UCl<sub>5</sub>. Miscible with ethanol, ether, benzene, chloroform, pyridine, etc. Immediately decomposed in water.

REFERENCE:

R. G. Jones, E. Bindschadler, G. Karmas, F. A. Yoeman and H. Gilman. J. Amer. Chem. Soc. <u>78</u>, 4287 (1956).

# Uranium (VI) Ethoxide

# U(OC<sub>2</sub>H<sub>5</sub>)<sub>6</sub>

#### $U(OC_2H_5)_5 + NaOC_2H_5 = NaU(OC_2H_5)_6$

 $2 \text{ NaU}(\text{OC}_{2}\text{H}_{5})_{6} + (\text{C}_{6}\text{H}_{5}\text{CO})_{2}\text{O}_{2} = 2 \text{ U}(\text{OC}_{2}\text{H}_{5})_{6} + 2 \text{ C}_{6}\text{H}_{5}\text{COONa}$ 

A 500-ml. three-neck flask is fitted with a gas-tight stirrer and a gas outlet tube. This flask is used to prepare a sodium ethoxide solution from 300 ml. of absolute alcohol and 1.69 g. (0.074 g.-atoms) of Na. When all of the Na is dissolved, the solution is cooled under N<sub>2</sub> and 20 ml. (34 g., 0.074 moles) of uranium (V) ethoxide (see above for preparation) is added with a pipette. The brown color of the ethoxide disappears, and a clear, light-green solution of NaU(OC<sub>2</sub>H<sub>5</sub>)<sub>6</sub> is formed. Now, 8.90 g. (0.037 moles) of dry benzoyl peroxide powder is added in three equal portions at about 10-minute intervals while vigorously stirring. The mixture becomes warm, acquiring a red color and forming a gelatinous precipitate of sodium benzoate. After one hour of additional stirring under N<sub>2</sub>, half of the material is transferred (still under  $N_2$ ) to a 250-ml. Claisen flask fitted with a distilling condenser, and the alcohol is distilled out at atmospheric pressure on a steam bath. The second half is then added and the procedure repeated.

The receiver flask is now heated on an oil bath and the contents are subjected to a vacuum distillation. At first, with the bath temperature as high as 140 °C and at 5-10 mm., the last of the alcohol is removed; the uranium (VI) ethoxide distills out at the same bath temperature but at a high vacuum (0.003 mm.). Yield: 20 g. Rectification in high vacuum affords the pure product. Yield: 16 g. (43%). The boiling point is 72-74 °C at 0.001 mm.

SYNONYM:

Uranium (VI) ethylate.

PROPERTIES:

Red, mobile liquid; d 1.563. Monomeric in benzene. Readily soluble in benzene, ether, petroleum ether, etc. Extremely sensitive to moisture; forms uranyl hydroxide when hydrolyzed. Strong oxidizing agent. Readily reduced to uranium (V) ethoxide. Synproportionates with U (IV) ethoxide according:

$$U(OC_{2}H_{5})_{6} + U(OC_{2}H_{5})_{4} = 2 U(OC_{2}H_{5})_{5}$$

REFERENCE:

R. G. Jones, E. Bindschadler, D. Blume, G. Karmas, G. A. Martin, J. R. Thirtle, F. A. Yoeman and H. Gilman, J. Amer. Chem. Soc. <u>78</u>, 6030 (1956).

#### Uranyldibenzoylmethane

 $UO_2(C_{15}H_{11}O_2)_2$ 

Methanolic solutions of uranyl acetate and dibenzoylmethane are combined in the cold. An intense reddish-yellow color appears at once, and after a few seconds uranyldibenzoylmethane powder begins to separate; it can be recrystallized from a large amount of hot alcohol. Other solvents may also be used for purification; however, one must bear in mind that uranyldibenzoylmethane forms well-crystallized addition compounds with almost all solvents.

Used analytically for the rapid separation of rare earths produced in the fission of uranium, since these do not form complexes with dibenzoylmethane in the presence of water. The uranium can be rapidly and conveniently separated by extraction (as the  $UO_2$ complex).

PROPERTIES:

Orange-red crystals, which change color at about 180°C and begin to decompose at 245°C. Readily soluble in all ketones and esters and in pyridine; moderately soluble in ethyl alcohol; sparingly soluble in ether; insoluble in hydrocarbons such as benzene, toluene and naphtha. Stable to water, but is decomposed by acids and alkalies (even by ammonia).

REFERENCES:

H. Götte. Z. Naturforsch. <u>1</u>, 378 (1946). Preparation of dibenzoylmethane: A. Magnani and S. M. McElvain. Org. Syntheses, collective Vol. 3, p. 251, New York-London, 1955.

# SECTION 25

Manganese

H. LUX

#### Manganese

#### Mn

#### I. BY ELECTROLYSIS

Very pure Mn may be produced by electrolysis under the following conditions:

The electrolysis is performed with anode and cathode spaces separated from each other, using canvas or a ceramic substance as the cell diaphragm. The cathode electrolyte contains 70 g. of  $MnSO_4 \cdot 4H_2O$  and 200 g. of  $(NH_4)_2SO_4$  per liter, the anode electrolyte 100 g. of  $(NH_4)_2SO_4$  per liter. The cathode is a polished stainless steel sheet; the anode is a lead sheet. The cathodic current density is 0.16 amp./in.<sup>2</sup> and the temperature of the electrolyte should not exceed 40°C. The pH value in the cathode cell should be maintained between 4.5 and 8.5, and the free sulfuric acid content of the anode cell should not exceed 5%. To prevent oxidation of the catholyte and to promote uniform deposition of Mn, a small quantity of a saturated  $SO_2$  solution is added from time to time to the catholyte so that a concentration of about 0.1-1 g. of  $SO_2$ /liter is maintained in it.

The current efficiency is about 50-70%. In addition to impurities, the content of which is a function of the purity of the starting electrolyte, the metal contains up to 0.02% S and some H<sub>2</sub>; however, the latter can be readily removed by heating in vacuum. The  $\gamma$ -Mn product is silver-gray, polishes readily and is stable in air. The boundary layer in contact with the cathode shows gradual transition to fine-grained  $\beta$ -Mn and is therefore hard. If performed under different conditions, the electrolysis will produce shiny layers of metal which rapidly turn dark upon exposure to air; in this case the metal should be immersed in a 5% Na<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub> solution immediately upon removal from the electrolyte; this treatment passivates it and permits it to retain its lustrous surface.

#### **II.** BY DISTILLATION

Very pure Mn can be obtained by the distillation of Mn prepared via the aluminothermic or the electrolytic methods. The metal, in a sintered  $Al_2O_3$  boat, is placed in a ceramic tube closed at one end. The tube must be pretested for gas tightness, and a vacuum of at least 0.005 mm. Hg should be established in it. The tube is heated in a Globar furnace to a temperature of 1250 to 1350°C, at which temperature the Mn vapor pressure is 1-2 mm. The distilled metal deposits as small needles on a tubular nickel sleeve cooled by running cold water and located in the vicinity of the boat; the metal can be loosened by slight tapping. The product is extremely reactive and ignites upon exposure to air; all subsequent handling must therefore take place in an Ar atmosphere in the absence of  $O_2$ .

Manganese prepared under the same conditions but deposited on an uncooled surface, e.g., an alumina rod, is less reactive. The dense, silvery scales of the crystalline metal are easily stripped off and reduced to powder. This product is the  $\alpha$ -modification, which is stable at temperatures below 742 °C.

#### PROPERTIES:

Atomic weight 54.93. M.p. 1212°C, b.p. 2152°C. Electrolytically precipitated Mn: d 7.2, crystal structure  $\gamma$ -Mn, A 6 type. Distilled Mn: d 7.44, crystal structure  $\alpha$ -Mn, A12 type.

REFERENCES:

- I. R. Springer. Die elektrolytische Abscheidung des Mangans [The Electrolytic Deposition of Manganese], Akad. Verlagsges., Leipzig, 1951; S. M. Shelton and M. B. Royer. Trans. Electrochem. Soc. <u>74</u>, 447 (1938), Chem. Zentr. <u>1939</u>, I, 2284; I. A. Mendelev, S. I. Orlova and Y. S. Shpichinetskiy. Tsvet. Metal. <u>16</u>, 53 (1941), Chem. Zentr. <u>1942</u>, II, 2196; E. Herrmann. Ann. Physik [5] <u>21</u>, 139 (1934).
- II. R. Schenk and A. Kortengråber. Z. anorg. allg. Chem. <u>210</u>, 273 (1933); H. Haraldsen and W. Klemm. Ibid. <u>220</u>, 184 (1934);
  M. L. V. Gayler. Metallwirtschaft <u>9</u>, 678 (1930); M. Picon and J. Flahaut. Comptes Rendus. Hebd. Séances Acad. Sci. <u>237</u>, 569 (1953).

# Manganese (II) Oxide

# MnO

 $\begin{array}{l} Mn(NO_3)_2 = MnO_2 + 2 NO_2; \ MnO_2 + H_2 = MnO + H_2O \\ (\cdot 6 H_2O) \\ 287.0 \end{array} \end{array}$ 

The oxide varies from grassy green to light green and may be obtained from any of the oxides (or other suitable salts of manganese) by reduction with  $H_2$  at temperatures below 1200°C. Thus, for example,  $Mn(NO_3)_2 \cdot 6H_2O$  is heated in air to about 300°C; the product (approx.  $MnO_{1.95}$ ) is ground to powder and reduced for 4 hours at 800°C with pure, oxygen-free hydrogen; the reaction rate becomes appreciable about 450°C. With prolonged heating or high temperatures the product turns an increasingly grayish color and finally becomes light gray; in the presence of oxygen, it is brown. A reduction temperature of 800°C is sufficient for hydrated oxides;  $MnCO_3$  must be heated to 1000-1100°C for 15-20 minutes.

PROPERTIES:

M.p. 1785°C; d5.18. Crystal structure: B1 (rock salt) type.

REFERENCES:

P. Dubois. Ann. Chimie [11] <u>5</u>, 432; for preparation by heating MnCO<sub>3</sub> or MnC<sub>2</sub>O<sub>4</sub> in a high vacuum, see M. LeBlanc and G. Wehner. Z. phys. Chem. A <u>168</u>, 61 (1934); T. E. Moore, M. Ellis and P. W. Selwood. J. Amer. Chem. Soc. <u>72</u>, 856 (1950); for preparation by heating MnCO<sub>3</sub> in flowing N<sub>2</sub>, see H. Ulich and H. Siemonsen. Arch. Eisenhüttenwesen <u>14</u>, 27 (1940); Z. Elektrochem. 45, 637 (1939).

#### Manganese (II) Hydroxide

#### Mn(OH)2

$$\begin{array}{c} MnCl_2 + 2 \text{ KOH} = Mn(OH)_2 + 2 \text{ KCl} \\ \frac{(\cdot 4 \text{ H}_2 \text{O})}{197.9} & 112.2 & 89.0 & 149.1 \end{array}$$

In the method of Simon, a solution of 300 g. of analytically pure KOH in 500 ml. of water in a round-bottom flask (see Fig. 327) is heated for about one half hour while a stream of completely  $O_2$ -free hydrogen is passed through; a completely  $O_2$ -free solution of 15-17 g. of MnCl<sub>2</sub> · 4 H<sub>2</sub>O in 15 ml. of boiled water is then added from a dropping funnel. The mixture is then heated to 190-200°C (as rapidly as possible) on an oil bath while H<sub>2</sub> is bubbled through. When the amorphous precipitate of Mn(OH)<sub>2</sub> is completely dissolved, the flask is allowed to cool slowly on the oil bath, whereby the compound precipitates out as white flakes of pearly sheen.

After cooling to room temperature, boiled water prepurged with  $H_2$  is added from a dropping funnel until the flask is almost filled; the liquor is then siphoned off by means of a glass tube which

reaches almost to the bottom of the flask (the flow of  $H_2$  should not be interrupted either during this or the preceding operation). The glass tube is connected to a Pyrex glass filter (constantly flushed with a  $H_2$  stream), which in turn is attached to a suction flask; the latter is connected to the suction pump by way of a wash

bottle containing a solution of  $CrCl_2$ . The addition of small amounts of water (washing operation) to the flask is repeated several times. The crystals are transferred (by shaking the flask) to the glass filter, washed on the filter, first with a large quantity of  $O_2$ -free water (a second dropping funnel is used), then with absolute alcohol through which H<sub>2</sub> is bubbled, and finally with peroxide-free ether. The product, still on the glass filter, is then dried in a desiccator over P<sub>2</sub>O<sub>5</sub> while maintaining a high vacuum.

This procedure gives a moderate yield of a well-crystallized product. Larger amounts of the microcrystalline substance are prepared more conveniently by the method of Scholder and Kolb [boiling with concentrated sodium hydroxide to which  $(NH_3OH)Cl$  is added].



Fig. 327. Preparation of manganese(II) hydroxide.

PROPERTIES:

When free of alkaline hydroxides and amorphous components, the dry product can be kept in an air-filled desiccator for weeks. The crystalline compound occurs in nature as pyrochroite.

Solubility (18°C) 0.0019 g./liter. d 3.258. Crystal structure; type C 6.

**REFERENCES:** 

A. Simon. Z. anorg. allg. Chem. <u>232</u>, 369 (1937); T. E. Moore, M. Ellis and P. W. Selwood. J. Amer. Chem. Soc. <u>72</u>, 858 (1950);
 R. Scholder and A. Kolb. Z. anorg. allg. Chem. <u>264</u>, 211 (1951).

# Manganese (III) Oxide

# $\gamma$ -Mn<sub>2</sub>O<sub>3</sub>, $\gamma$ -MnO(OH)

 $2 \text{ MnSO}_4 \cdot 4 \text{ H}_2\text{O} + \text{ H}_2\text{O}_2 + 4 \text{ NH}_3 = 2 \text{ MnO}(\text{OH}) + 2 (\text{NH}_4)_2\text{SO}_4 + 6 \text{ H}_2\text{O}$ 446.1 34.0 68.0 175.9 264.3

The method of Marti gives  $\gamma$ -MnO(OH) with a well-defined x-ray pattern. A solution of 2.2 g. of MnSO<sub>4</sub> · 4 H<sub>2</sub>O (10 mmoles)

in 350 ml. of water is treated in a large beaker (very vigorous mechanical agitation) with 34 ml. of a 3%  $H_2O_2$  solution (30 mmoles). With continued very vigorous agitation, 50 ml. of a 0.2 M NH<sub>3</sub> solution (10 mmoles) is added at once from a graduated cylinder. The dark-brown or black suspension, which is evolving oxygen, is brought to a boil as rapidly as possible. The boiling is continued for about 4 minutes and the solution is filtered, washed with 1.5 liters of hot water, and dried over  $P_2O_5$  in vacuum at temperatures below about 100 °C.

Careful dehydration of  $\gamma$ -MnO(OH) (vacuum, 250°C) yields  $\gamma$ -Mn<sub>2</sub>O<sub>3</sub>.  $\gamma$ -MnO(OH) occurs in nature as manganite.

REFERENCES:

W. Marti. Uber die Oxidation von Manganhydroxyd und über höherwertige Oxyde und Oxydhydrate des Mangans [The Oxidation of Manganese Hydroxide and the Higher Oxides and Hydrated Oxides of Manganese], Thesis, Univ. of Bern, 1944, p. 83;
W. Feitknecht and W. Marti. Helv. Chim. Acta <u>28</u>, 142 (1945);
T. E. Moore, M. Ellis and P. W. Selwood. J. Amer. Chem. Soc. <u>72</u>, 861 (1950); P. Dubois. Ann. Chimie [11] <u>5</u>, 434 (1936);
A. Simon and S. Fehér. Z. Electrochem. <u>38</u>, 137 (1932);
F. Krull. Z. anorg. allg. Chem. <u>208</u>, 134 (1932); K. L. Orr. J. Amer. Chem. Soc. <u>76</u>, 857 (1954).

# Manganese (IV) Oxide

#### MnO<sub>2</sub>

$$\begin{array}{l} Mn(NO_3)_2 \cdot 6 H_2O = MnO_2 + 2 NO_2 + 6 H_2O \\ 287.0 \qquad 86.9 \end{array}$$

The starting  $Mn(NO_3)_2 \cdot 6H_2O$  is decomposed in air by heating to about 190°C; the product is ground to powder, boiled with nitric acid (conc. HNO<sub>3</sub> diluted 1:6) and heated in air to 450-500°C. The x-ray pattern of the product clearly shows the lines of pyrolusite ( $\beta$ -MnO<sub>2</sub>). At atmospheric pressure, oxygen begins to split off above 530°C in air, and above 565°C in oxygen.

Alternate methods: a) From  $MnCl_2$  and  $(NH_4)_2S_2O_8$  in aqueous solution [A. Simon and F. Fehér, Z. Elektrochem. 38, 137 (1942)].

b) From  $NH_4MnO_4$  and  $NH_3$  in aqueous solution (A. Harzer, German Patent 713,904, Class 12, Group 3, as well as the references cited below).

c) From Mn<sub>2</sub>O<sub>7</sub> [P. Dubois, Ann. Chimie [11] <u>5</u>, 411 (1936); A. Simon and F. Fehér, Kolloid-Z. <u>54</u>, 50 (1931); A. Simon and F. Fehér, Z. Elektrochem. 38, 137 (1932)]. SYNONYM:

Manganese dioxide.

**REFERENCES:** 

C. Drücker and R. Hüttner. Z. phys. Chem. <u>131</u>, 263 (1928); O. Glemser, Ber. dtsch. chem. Ges. <u>72</u>, 1879 (1939); F. Krüll. Z. anorg. allg. Chem. <u>208</u>, 134 (1932); W. Marti. Thesis, Univ. of Bern, 1944; T. E. Moore, M. Ellis and P. W. Selwood. J. Amer. Chem. Soc. <u>72</u>, 863(1950); G. Butler and H. R. Thirsk. J. Electrochem. Soc. <u>100</u>, 297 (1953); G. Gattow and O. Glemser. Z. anorg. allg. Chem. <u>309</u>, 121 (1961).

# Manganese (VII) Oxide

#### Mn<sub>2</sub>O<sub>7</sub>

 $\begin{array}{l} 2 \text{ KMnO}_4 + 2 \text{ H}_2 \text{SO}_4 = 2 \text{ KHSO}_4 + \text{ Mn}_2 \text{O}_7 + \text{ H}_2 \text{O} \\ 316.1 \quad 196.2 \quad 272.3 \quad 221.9 \end{array}$ 

Concentrated  $H_2SO_4$  (15 ml., d 1.84) is placed in a dry porcellain mortar precleaned with chromosulfuric acid; then 23 g. of KMnO<sub>4</sub> is carefully added over a period of 10-15 minutes, while constantly stirring with a pestle. To obtain the desired result, the following precautions must be observed: only very pure KMnO<sub>4</sub> crystals, free of dust and organic substances (preferably Merck A.R. quality; do not reduce the crystals to powder), can be used. The reaction slurry should be left standing overnight in a dry spot, protected against dust. Porous pyrolusite is formed during this time, and the  $Mn_2O_7$  oil is very gently kneaded out from it. Proper safety measures must be observed during the preparation and further workup of the material, since it often explodes for no apparent reason (an asbestos face shield with safety glasses and heavy leather gloves should be worn and one should work behind heavy glass plate and a fine wire screen).

Yield: 10 g. (62%). The product is entirely free of  $K^+$  and  $SO_4^{2-}$ .

SYNONYMS:

Manganese heptoxide, permanganic acid anhydride.

PROPERTIES:

An oil with green metallic luster in reflected light; dark red in transmitted light; specific odor. M.p.  $5.9^{\circ}C$ ;  $d_4^{2\circ}$  2.396; heat

of formation: -177.4 kcal (20°C); dissociation at approx. 55°C, detonation at 95°C. In vacuum, rapid and explosive dissociation above 10°C. Forms  $\alpha$  -Mn<sub>2</sub>O<sub>3</sub> during explosive decomposition,  $\gamma$ -MnO<sub>2</sub> during slow dissociation. Soluble in conc. H<sub>2</sub>SO<sub>4</sub> and H<sub>3</sub>PO<sub>4</sub> with an olive green color. Hygroscopic; dissociates slowly in humid air to MnO<sub>2</sub>, liberating O<sub>3</sub>-containing O<sub>2</sub> and, occasionally, a red mist of HMnO<sub>4</sub>. Stable under refrigeration (-10°C) provided anhydrous conditions are maintained. Reacts explosively with most organic compounds; attacks acetic acid, acetic anhydride and CCl<sub>4</sub> even below room temperature. Dangerous compound ! The impact sensitivity of Mn<sub>2</sub>O<sub>7</sub> is equal to that of mercuric fulminate.

REFERENCES:

J. M. Lovén. Ber. dtsch. chem. Ges. <u>25</u>, Ref. 620 (1892); A. Simon and F. Fehér. Z. Elektrochem. <u>38</u>, 138 (1932); O. Glemser and H. Schröder. Z. anorg. allg. Chem. <u>271</u>, 294 (1953).

# Sodium Manganate (V)

#### $Na_3MnO_4 \cdot 0.25 NaOH \cdot 12 H_2O$

$$2 \text{ KMnO}_4 + 2 \text{ Na}_2\text{SO}_3 \cdot 7 \text{ H}_2\text{O} + 6 \text{ NaOH} + 4 \text{ H}_2\text{O}$$

$$316.1 \quad 504.3 \quad 240.0 \quad 72.1$$

$$= 2 \text{ Na}_3\text{MnO}_4 \cdot 10 \text{ H}_2\text{O} + 2 \text{ Na}_2\text{SO}_4 + 2 \text{ KOH}$$

$$736.2 \quad 284.1 \quad 112.2$$

A solution of 2 g. of very fine  $\text{KMnO}_4$  powder in 50 ml. of 28% sodium hydroxide is triturated in a small Erlenmeyer flask with 3.5 g. of finely divided  $\text{Na}_2\text{SO}_3 \cdot 7\text{H}_2\text{O}$ ; the flask stands in an ice bath. The trituration requires about 10 minutes, that is, until a light-blue crystalline slurry is obtained. This is then transported by vacuum onto an ice-cooled glass filter, and the product washed thoroughly with 28% sodium hydroxide at 0°C. The wet preparation is rapidly spread in a thin layer on fresh clay and stored at 0°C in an evacuated desiccator (no drying agent). The salt has the stoichiometric composition and contains, in addition to the hydroxide, about 0.4% SO<sub>3</sub>; the preparative procedure should be designed to keep contamination by silicates or aluminates to a minimum.

#### PROPERTIES:

Formula weight 368.1. The salt, in the form of well-crystallized sky-blue rods, remains stable at 0°C if kept free of  $H_2O$ 

and  $CO_2$ . Solubility in 28% NaOH at 0°C is equivalent to 0.06%  $Mn_2O_5$ . A solution of the salt in 50% potassium hydroxide turns grassy green upon heating or dilution; simultaneously,  $MnO_2$  is precipitated according to the equation:

$$2 \text{Na}_3 \text{MnO}_4 + 2 \text{H}_2 \text{O} = \text{Na}_2 \text{MnO}_4 + \text{MnO}_2 + 4 \text{NaOH}.$$

REFERENCES:

H. Lux. Z. Naturforschg. <u>1</u>, 281 (1946) and unpublished work. The preparation of the sulfate-free product is described by R. Scholder, D. Fischer and H. Waterstradt, Z. anorg. allg. Chem. <u>277</u>, 236 (1954).

#### Potassium Manganate (VI)

#### K<sub>2</sub>MnO<sub>4</sub>

$$2 \text{ KMnO}_4 + 2 \text{ KOH} = 2 \text{ K}_2 \text{MnO}_4 + \frac{1}{2} \text{ O}_2 + \text{ H}_2 \text{O}_3 \text{ I}_{6.1} \qquad 112.2 \qquad 394.2$$

A solution of 30 g. of KOH in 50 ml. of water is prepared; 10 g. of KMnO<sub>4</sub> is added and the mixture is boiled in an open 250-ml. Erlenmeyer flask until a pure green solution is obtained. The water lost by evaporation is then replaced and the flask set in ice. The precipitated black-green crystals, which show a purplish luster, are collected on a Pyrex glass filter, washed (high suction) with some 1N potassium hydroxide, and dried over  $P_2O_5$ . The salt can be recrystallized by dissolving in dil. potassium hydroxide and evaporating in vacuum.

PROPERTIES:

Formula weight 197.1. Solubility (20°C) in 2N potassium hydroxide 224.7 g./liter, in 10N potassium hydroxide 3.15 g./liter.

REFERENCES:

K. A. Jensen and W. Klemm. Z. anorg. allg. Chem. <u>237</u>, 47 (1938);
R. Luboldt. J. prakt. Chem. <u>77</u>, 315 (1859). Preparation of an especially pure, KOH-free product is described by R. Scholder and H. Waterstradt, Z. anorg. allg. Chem. <u>277</u>, 172 (1954).

#### Barium Manganate (VII)

#### Ba(MnO<sub>4</sub>)<sub>2</sub>

A solution of 100 g. of  $KMnO_4$  and 100 g. of  $Ba(NO_3)_2$  in 1.5 liters of boiling water is prepared and treated with 20 g. of  $Ba(OH)_2$ .  $8 H_2O$ . The solution is heated on a water bath with frequent agitation until the evolution of  $O_2$  largely ceases, whereupon another 20 g. of  $Ba(OH)_2 \cdot 8 H_2O$  is added and the water lost by evaporation replaced. The procedure is continued until the liquid becomes colorless. When the sparingly soluble  $BaMnO_4$  settles out (together with some  $MnO_2$  and  $BaCO_3$ ), the liquid is decanted, the precipitate washed repeatedly with several liters of boiling water, boiled with a dilute solution of  $Ba(OH)_2$ , and rewashed thoroughly with boiling water.

The precipitate is then suspended in 1 liter of water and completely decomposed by introducing simultaneously  $CO_2$  and superheated steam. This takes a few hours. The solution is left to settle; the liquor is suction-filtered through glass and concentrated until the almost black crystals appear upon cooling. The yield is 65-80 g. (80 to 100% of theory). Permanganates of all types of metals can be prepared by reacting the Ba(MnO<sub>4</sub>)<sub>2</sub> with an equivalent quantity of sulfate.

SYNONYM:

Barium permanganate.

PROPERTIES:

Sparingly soluble in water; d 3.77.

**REFERENCES:** 

W. Muthmann. Ber. dtsch. chem. Ges. <u>26</u>, 1017(1893); H. G. Grimm,
 C. Peters and H. Wolff. Z. anorg. allg. Chem. <u>236</u>, 73 (1938).

#### Silver Manganate (VII)

#### AgMnO<sub>4</sub>

A hot (80°C) solution of 5 g. of  $AgNO_3$  in 100 ml. of water is added to a hot (80°C) solution of 4.66 g. of  $KMnO_4$  in 300 ml. of water to which a drop of conc. nitric acid has been added. The mixture is permitted to cool. Since the product still contains some K, it is recrystallized from water by slow cooling from 80°C.

The black, lustrous, needle-shaped crystals tend to decompose on prolonged storage. Because of its limited solubility, this salt is less suitable as a raw material for other permanganates than  $Ba(MnO_4)_2$ .

SYNONYM:

Silver permanganate.

PROPERTIES:

Solubility (room temperature) 9 g./liter  $H_{2}O$ ; d 4.49. Crystal structure: type  $HO_{2}$ .

REFERENCE:

W. Büssem and K. Herrmann. Z. Kristallogr. A 74, 459 (1930).

# BaSO<sub>4</sub>-KMnO<sub>4</sub> Solid Solution

The solid solution (mixed crystals), described and examined in detail by Grimm and Wagner, is prepared simply by mixing together solutions of  $Ba(NO_3)_2$  and  $K_2SO_4$ , both containing a high percentage of KMnO<sub>4</sub>. As an example, the following conditions were found to be suitable:

A solution of 1.31 g. (5 mmoles) of Ba(NO<sub>3</sub>)<sub>2</sub> and 50 g. of KMnO<sub>4</sub> in 1 liter of water is prepared; similarly, 0.87 g. (5 mmoles) of  $K_2SO_4$  and 50 g. of KMnO<sub>4</sub> are dissolved in 1 liter of water. Heating is required in both cases. The clear solutions—suction-filtered through glass, if necessary—are brought to 50°C, added together, and allowed to stand for a short time at 50°C. The crystals are then separated by suction filtration.

Washing the mixed crystals with acetone until the wash liquid turns a light rose results in a product with a  $KMnO_4$  content of 25-30 mole %; treatment with water, however, readily decomposes

the crystals. A more stable solid solution, containing 6-8 mole % KMnO<sub>4</sub>, is obtained by washing with water at 50°C, rinsing with a SO<sub>2</sub> solution, and repeating the washing. It is possible to boil the rose to purple powder with solutions of SO<sub>2</sub> or other reducing agents without a change in composition. The surface of the compound is decomposed on prolonged exposure to sunlight (with precipitation of manganese oxides).

REFERENCES:

H. G. Grimm and G. Wagner. Z. phys. Chem. <u>132</u>, 135 (1928); see also A. Benrath and H. Schackmann. Z. anorg. allg. Chem. <u>218</u>, 139 (1934).

# Potassium Manganese (III) Chloride

# K2MnCl5

In the method of Weinland and Dinkelacker the compound is prepared as follows: 5 g. of KMnO<sub>4</sub> powder is added (constant shaking) to 50 ml. of approx. 40% HCl (d 1.19). The initial fine, brown precipitate is slowly dissolved on frequent shaking, while copious quantities of  $Cl_2$  are being evolved. The solution is left standing for two hours, then decanted from any black K<sub>2</sub>MnCl<sub>6</sub> that may have precipitated, and conc. aqueous KCl is added dropwise to the deep dark-red to brown solution (constant agitation) until the liquid becomes nearly colorless. The crystalline, brownish K<sub>2</sub>MnCl<sub>5</sub> precipitate is filtered off by suction and dried over KOH.

REFERENCES:

R. F. Weinland and P. Dinkelacker. Z. anorg. allg. Chem. <u>60</u>, 173 (1908). For the hydrate K<sub>2</sub>MnCl<sub>5</sub> · H<sub>2</sub>O, see C. E. Rice. J. Chem. Soc. (London) <u>73</u>, 260 (1898).

# Potassium Hexachloromanganate (IV)

# K2MnCl6

 $\begin{array}{rl} Ca(MnO_4)_2 + 16 \ HCl + 4 \ KCl = 2 \ K_2MnCl_8 + CaCl_2 + 8 \ H_2O + 3 \ Cl_2 \\ 277.9 & 583.5 & 298.2 & 691.7 & 111.0 \end{array}$ 

In the method of Weinland and Dinkelacker, 5.0 g. of  $Ca(MnO_4)_2$ is added (constant agitation) to 50 ml. of 40% hydrochloric acid cooled with an ice-salt mixture. A solution of 2 g. of KCl in 8 ml. of water is added simultaneously in drops. The almost black, crystalline precipitate is rapidly separated by suction filtration and dried for a short time on a clay plate over conc.  $H_2SO_4$ .

PROPERTIES:

Formula weight 345.9. Small, translucent, deep dark-red crystals; liberates  $Cl_2$  continuously even in dry air.

REFERENCE:

R. F. Weinland and P. Dinkelacker. Z. anorg. allg. Chem. <u>60</u>, 173 (1908).

# Manganese (II) Sulfide

#### MnS

a-MnS, GREEN, CUBIC

This modification, which has been thoroughly studied by x-ray techniques, is obtained via the method of Classen.

A boiling solution of about 10 g. of  $MnCl_2 \cdot 4H_2O$  in 500 ml. of water containing a small quantity of  $K_2C_2O_4$  is reacted with an excess of a 50% NH<sub>3</sub> solution and saturated at its boiling point with H<sub>2</sub>S. Upon further heating, the initial flesh-colored MnS precipitate is rapidly converted to the stable dark-green  $\alpha$ modification. To remove any coprecipitated sulfur, the sulfide is boiled three times with a dil. solution of freshly prepared, colorless (NH<sub>4</sub>)<sub>2</sub>S and, after filtering, washed successively with H<sub>2</sub>S-containing water, alcohol and ether. It is dried in an oil-pump vacuum at 120°C.

The dry preparation of  $\alpha$ -MnS(alabandite) is described by H. Haraldsen and W. Klemm, Z. anorg. allg. Chem. <u>220</u>, 271 (1936); for the synthesis of MnS<sub>2</sub> (hauerite), see W. Biltz and F. Wiechmann, ibid. <u>228</u>, 271 (1936).

PROPERTIES:

Formula weight 86.99. M.p. 1610°C; d 3.99. Crystal structure: B1 (rock salt) type.
**REFERENCES:** 

 H. Schnaase. Z. phys. Chem. (B) <u>20</u>, 89 (1933); F. Mehmed and H. Haraldsen. Z. anorg. allg. Chem. <u>235</u>, 194(1938); A. Classen. Z. analyt. Chem. 16, 319 (1877).

 $\beta$ -MnS, RED, CUBIC

In the method of Schnaase, the  $\gamma$ -modification is obtained by introducing H<sub>2</sub>S into a cold solution of 50 g. of Mn(CH<sub>3</sub>COO)<sub>2</sub> · 4 H<sub>2</sub>O in 300 ml. of water. After some time, most of the sulfide settles on the bottom as a reddish-brown precipitate, while another fraction adheres to the glass wall as a beautiful, miniumred scale. The precipitate is washed with H<sub>2</sub>S-saturated water (the preferred washing method is decantation), filtered off with suction while H<sub>2</sub>S is being passed over it, washed again with alcohol and ether, and dried in an oil-pump vacuum at 80°C.

Crystal structure: B3 (sphalerite) type.

**REFERENCES:** 

 H. Schnaase. Z. phys. Chem. (B) <u>20</u>, 89 (1933); F. Mehmed and H. Haraldsen. Z. anorg. allg. Chem. <u>235</u>, 194 (1938).

 $\gamma$ -MnS, RED, HEXAGONAL

Schnaase prepares the  $\gamma$ -modification by first dissolving 20 g. of analytically pure MnCl<sub>2</sub> • 4 H<sub>2</sub>O and some NH<sub>4</sub>Cl in 500 ml. of boiled water through which a stream of O<sub>2</sub>-free N<sub>2</sub> is being bub-Then H<sub>a</sub>S is introduced at the boiling temperature and bled. Mn(OH)<sub>2</sub> is precipitated out with a slight excess of concentrated NH<sub>3</sub> solution. The precipitate is initially white, gradually turns a light pink upon further contact with H<sub>2</sub>S, and finally assumes the color of red meat, while the sulfide forming the surface layer is first orange yellow and later turns vermilion red. After settling, the precipitate is washed twice by decanting with H<sub>2</sub>S-saturated water and boiled for two days in a 10% NH<sub>3</sub> solution while H<sub>2</sub>S is bubbled through. Finally it is washed by decanting several times with HaS-saturated water, filtered off in the absence of air while under a N<sub>2</sub> stream, washed with alcohol and ether, and dried in an oil-pump vacuum at 80°C. Any coprecipitated sulfur is removed by extraction with boiling CS<sub>2</sub> under nitrogen.

Crystal structure: B4 (wurtzite) type.

The conversion of the two dry, metastable red MnS modifications to the stable form starts at 200°C. The rate is appreciable, and is higher at 300°C. The red modifications also differ from the  $\alpha$ -form in their magnetic behavior. REFERENCES:

 H. Schnaase. Z. phys. Chem. (B) <u>20</u>, 89 (1933); F. Mehmed and H. Haraldsen. Z. anorg. allg. Chem. <u>235</u>, 194 (1938).

## Manganese (III) Sulfate

### $Mn_2(SO_4)_3$

 $2 \text{ KMnO}_4 + 4 \text{ H}_2 \text{SO}_4 = \text{Mn}_2 (\text{SO}_4)_3 + \text{K}_2 \text{SO}_4 + 4 \text{ H}_2 \text{O} + 2 \text{ O}_2$ 316.1 392.3 398.0 174.3 72.1

In the method of Domange, the crystalline  $Mn_2(SO_4)_3$  is prepared by introducing 20 g. of KMnO<sub>4</sub> powder into 100 ml. of  $H_{2}SO_{4}$  (d 1.84) contained in a porcelain crucible (agitation); the mixture is carefully heated for 10 minutes at 60°C while stirring vigorously, whereby vapors of the explosive Mn<sub>2</sub>O<sub>7</sub> are removed. The solution is then heated to 70°C (or at most to 75°C) with continued very vigorous agitation and accurate temperature control (thermometer). Vigorous evolution of  $O_2$  takes place while the liquid turns brown and becomes clouded, with a tendency for a spontaneous rise in temperature, so that the danger of an explosion persists. After about 15 minutes, with most of the reaction completed, the danger of an explosion passes. The mixture is now slowly (10 minutes) brought to 140°C with continued stirring and is finally raised to 200°C (in 15 minutes). Following slow cooling, the solution is washed twice by decantation with  $H_2SO_4$  (d 1.84) to remove the  $K_2SO_4$ . The product is collected on a glass filter and placed on a clay plate. The latter is placed for three to four days in a desiccator containing  $P_2O_5$ . The  $H_2SO_4$  may be completely removed by heating the preparation, together with a receiver cooled to -80 °C, for about three hours at 200°C and high vacuum; the salt itself does not begin to decompose until about 300°C.

#### PROPERTIES:

Extremely hygroscopic salt consisting of small, dark-green needles. Soluble in 75.25% wt. % or more  $H_2SO_4$  without alteration; a brown salt of composition  $Mn_2(SO_4)_3 \cdot H_2SO_4 \cdot 6H_2O$  crystallizes out from dilute (preferably 70%) sulfuric acid [A.R.J.P. Ubbelohde, J. Chem. Soc. (London) <u>1935</u>, 1605]. Sulfuric acid in concentrations lower than 52% produces hydrolysis.

#### REFERENCE:

L. Domange. Bull. Soc. Chim. France [5] 4, 594 (1937).

# Cesium Manganese (III) Sulfate CsMn(SO<sub>4</sub>)<sub>2</sub> · 12 H<sub>2</sub>O

 $Cs_{2}SO_{4} + 2 Mn(CH_{3}COO)_{3} \cdot 2 H_{2}O + 3 H_{2}SO_{4}$   $(^{1}/_{10}): 36.2 \qquad 53.6 \qquad 29.4$ 

 $= 2 \operatorname{CsMn}(\operatorname{SO}_4)_2 \cdot 12 \operatorname{H}_2 O + 6 \operatorname{CH}_3 \operatorname{COOH}_{119.2}$  36.0

In Christensen's method, 5.3 g. (0.01 mole) of  $Mn(CH_3COO)_3$ . 2H<sub>2</sub>O is dissolved in sulfuric acid (conc. H<sub>2</sub>SO<sub>4</sub> diluted 1:3), starting at room temperature. Then, a solution of 3.6 g. (0.01 mole) of  $Cs_2SO_4$  in 10 ml. of sulfuric acid of the same concentration is added; the solution is first cooled to -25°C in order to accelerate the precipitation and then left to stand for a long time at -5°C. The alum crystals are filtered off with suction and stored in a hermetically sealed bottle.

SYNONYM:

Cesium manganese alum.

PROPERTIES:

Coral-red crystalline powder. Melts at 40 °C in the water of hydration; however, turns brownish black slightly above room temperature. The hydrated  $Mn_2O_3$  is precipitated upon addition of water. The corresponding Rb alum melts at room temperature.

REFERENCES:

O. T. Christensen. Z. anorg. Chem. <u>27</u>, 329 (1901); H. Bommer. Z. anorg. allg. Chem. <u>246</u>, 281 (1941).

### Manganese Nitride

#### Mn<sub>4</sub>N

The nitride is prepared from very reactive sublimed manganese. The metal is heated in an apparatus which permits the continuous measurement of the amount of  $N_2$  used in the reaction. A completely  $O_2$ -free nitrogen is used, under a pressure of ~100 mm. and a temperature of 690°C. The reaction is continued until a constant final pressure is obtained (12-24 hours).

The product corresponds quite exactly to the formula  $Mn_4N$  (6.0% N;  $\epsilon$ -phase according to Hägg) and is strongly ferromagnetic. Below 400°C the homogeneous region of the phase extends from about 6.0 to 6.5% N.

**REFERENCES:** 

U. Zwicker. Z. Metallkunde <u>42</u>, 277 (1951); R. Schenk and A. Kortengräber. Z. anorg. allg. Chem. <u>210</u>, 273 (1933); G. Hägg. Z. phys. Chem. <u>B4</u>, 346 (1929); L. F. Bates, R. E. Gibbs and D. V. Reddi Pantulu. Proc. Phys. Soc. <u>48</u>, 665 (1936); see also H. Nowotny. Z. Elektrochem. <u>49</u>, 245 (1943).

For Mn<sub>4</sub>P, Mn<sub>2</sub>P, etc., see W. Blitz and F. Wiechmann, Z. anorg. allg. Chem. <u>234</u>, 117 (1937).

### Manganese (III) Acetate

#### $Mn(CH_3COO)_3$ , $Mn(CH_3COO)_3 \cdot 2H_2O$

#### Mn(CH<sub>3</sub>COO)<sub>3</sub> · 2 H<sub>2</sub>O

 $4 \operatorname{Mn}(\operatorname{CH_3COO})_2 \cdot 4 \operatorname{H_2O} + \operatorname{KMnO_4} + 8 \operatorname{CH_3COOH}$ <sup>1</sup>/<sub>10</sub>: 98.0 15.8 48.0  $= 5 \operatorname{Mn}(\operatorname{CH_3COO})_3 \cdot 2 \operatorname{H_2O} + \operatorname{CH_3COOK} + 10 \operatorname{H_2O}$ 134.0 9.8 18.0

In the method of Christensen, the salt is obtained in the following manner: 19.6 g. (80 mmoles) of Mn(CH<sub>3</sub>COO)<sub>2</sub> · 2H<sub>2</sub>O powder is added to 200 ml. of glacial acetic acid at the boiling temperature of the latter, and is stirred until completely dissolved. Then,  $KMnO_4$  powder (3.1 g. = 20 mmoles) is gradually added and the mixture heated for a short time with constant agitation. After cooling, 3 ml. of water is added to the darkbrown solution; the mixture is allowed to stand overnight. If the quantity of precipitate is too small, another 3 ml. of water is added and the solution stirred. The formation of crystallization nuclei is promoted by frequent rubbing of the container walls with a glass rod. As a rule, copious crystallization occurs within about one hour. If necessary, the solution is allowed to stand for a few more days (frequent agitation) until the mother liquor is almost colorless. The salt is then filtered off with suction, washed with some glacial acetic acid and recrystallized. The last procedure consists in dissolving 30 g. of salt in 200 ml. of glacial acetic acid (heating), filtering off and working up further in the manner described above. Finally, the salt is dried over CaO.

#### PROPERTIES:

The cinnamon-brown crystals of a silky luster are immediately decomposed by cold water (hydration).

In activity the compound is likely to be a complex salt with three nuclei, of the following structure:  $[Mn_3(CH_3COO)_6(H_2O)_2-(CH_3COO)_3 \cdot 4 H_2O]$ 

REFERENCES:

O. T. Christensen. Z. anorg. Chem. <u>27</u>, 325 (1901); R. F. Weinland and G. Fischer. Z. anorg. allg. Chem. 120, 161 (1921).

Mn(CH<sub>3</sub>COO)<sub>3</sub>

$$2 \operatorname{Mn}(\operatorname{NO}_3)_2 \cdot 6 \operatorname{H}_2 O + 15 (\operatorname{CH}_3 \operatorname{CO})_2 O$$

$${}^{1}_{/_{10}}: 57.4 153.0 \\ = 2 \operatorname{Mn}(\operatorname{CH}_3 \operatorname{COO})_3 + 4 \operatorname{NO}_2 + {}^{1}_{/_2} O_2 + 24 \operatorname{CH}_3 \operatorname{COOH}_{46.4} 144.1$$

Chrétien and Varga obtain the salt from  $Mn(NO_3)_2$  and acetic anhydride. A mixture of 20 g. of  $Mn(NO_3)_2 \cdot 6H_2O$  and 80 g. of acetic anhydride is heated slightly (shaking) until the vigorous, strongly exothermic reaction evolving large amounts of gas is well under way. When the reaction is completed, the homogeneous, oily liquid is cooled; the anhydrous acetate precipitates out as a brown, crystalline powder. The latter is collected on a glass filter, washed first with acetic anhydride and then with some ether to remove the odor of acetic acid, and stored in a closed container (anhydrous conditions). The yield, based on manganese, is 85%.

REFERENCE:

A. Chrétien and G. Varga. Bull. Soc. Chim. France [5] <u>3</u>, 2387 (1936).

### Potassium Trioxalatomanganate (III)

### $K_3[Mn(C_2O_4)_3] \cdot 3H_2O$

 $\begin{array}{c} \mathrm{KMnO_4} + 5 \ \mathrm{H_2C_2O_4} \cdot 2 \ \mathrm{H_2O} + \ \mathrm{K_2CO_3} = \mathrm{K_3}[\mathrm{Mn}(\mathrm{C_2O_4})_3] \cdot 3 \ \mathrm{H_2O} + 12 \ \mathrm{H_2O} \\ 158.0 \qquad \qquad 630.3 \qquad \qquad 138.2 \qquad \qquad 490.3 \\ & + \ 5 \ \mathrm{CO_2} \end{array}$ 

Cartledge and Ericks prepare the ferric ion-free compound (which is very sensitive to light) from analytically pure KMnO<sub>4</sub> according to the equation presented above. A solution of 31.5 g. (0.25 moles) of  $H_2C_2O_4 \cdot 2H_2O$  in 200 ml. of water is heated in

1470

a 500-ml. beaker to 70-75°C; then, 6.32 g. (0.04 moles) of KMnO<sub>4</sub> powder is added little by little (constant agitation) and as soon as the solution turns colorless, 6.9 g. (0.05 moles) of  $K_2CO_3$  is introduced in a similar manner. The mixture is cooled to 4-5°C (frequent stirring) and diluted with 150 ml. of 0-1°C water.

In all of the following operations light must be excluded as much as possible. The oxidation to  $Mn^{3+}$  is effected through the gradual addition of 1.58 g. (0.01 mole) of KMnO<sub>4</sub> powder; the solution is then stirred for about 10 minutes at 0 to 2°C. The intense cherry-red liquid is then suction-filtered through a glass filter precooled to 0°C, and is collected in a similarly cooled beaker. Next, the solution is reacted with half its volume of ice-cold alcohol and left to crystallize for two hours in an ice-salt mixture.

The precipitate is collected on a precooled glass filter, washed four times with 25 ml. of 50 vol. % alcohol, then with 95% alcohol, absolute alcohol and finally (three times) with ether; all of the wash liquids must be ice cold. After filtration with suction, the deep reddish-purple crystals are spread in a thin layer and exposed to air for a few hours; they are stored in brown bottles. The yield is ~50%.

PROPERTIES:

Very pure product can be stored for a long time at  $20^{\circ}$ C in the absence of air; stable for an almost unlimited time at  $-6^{\circ}$ C. Readily soluble in water: concentrated solutions are deep reddishbrown; very dilute or acidified solutions are yellowish-brown. The color change is due to the shift of the instantly established equilibrium:

$$[Mn(C_2O_4)_3]^{-3} + 2H_2O \rightarrow [Mn(C_2O_4)_2(H_2O)_2]^{-1} + C_2O_4^{-2}$$

The salt is a normal complex.

REFERENCE:

G. H. Cartledge and W. P. Ericks. J. Amer. Chem. Soc. <u>58</u>, 2061 (1936).

# Potassium Dioxalatodihydroxomanganate (IV) K<sub>2</sub>[Mn(C<sub>2</sub>O<sub>4</sub>)<sub>2</sub>(OH)<sub>2</sub>] · 2 H<sub>2</sub>O

$$\begin{array}{rcl} \mathrm{KMnO_4} &+& 3\,\mathrm{H_2C_2O_4} \cdot 2\,\mathrm{H_2O} &+& {}^{1}/{}_{2}\,\mathrm{K_2C_2O_4} \cdot \mathrm{H_2O} \\ 158.0 && 378.2 && 92.1 \\ && = & \mathrm{K_2[Mn(C_2O_4)_2(OH)_2]} \cdot 2\,\mathrm{H_2O} &+& 3\,\mathrm{CO_2} &+& 6{}^{1}/{}_{2}\,\mathrm{H_2O} \\ && & 379.2 \end{array}$$

This compound, discovered by Cartledge and Ericks, is prepared in a manner quite similar to that of  $K_3[Mn(C_2O_4)_3]$ . Thus, 17.64 g. (0.14 moles) of  $H_{2}C_{2}O_{4} \cdot 2H_{2}O$  is dissolved in 250 ml. of water and the solution is cooled to 0°C; then, 6.32 g. (0.04 moles) of KMnO<sub>4</sub> powder and 4.78 g. (0.026 moles) of  $K_{2}C_{2}O_{4} \cdot H_{2}O$  are added with constant agitation. The mixture is stirred vigorously for about 20 minutes; the temperature should rise gradually to 7°C during this time. As soon as  $CO_{2}$  begins to evolve at this temperature, the dark-green solution is rapidly cooled to 0°C in an efficient cooling mixture (swirling necessary) and is quickly suction-filtered through a Büchner funnel (filter paper). The filtrate is immediately placed in a cooling mixture and reacted with 100 ml. of alcohol in small portions at 0°C; the complex salt is thus precipitated as a very fine, crystalline powder. The latter is rapidly filtered off, washed successively with ice-cold 50% alcohol, 95% alcohol, absolute alcohol and ether, and stored at 0°C.

The salt can be recrystallized at  $0^{\circ}$ C: the powder is dissolved in 25 times its volume of cold 0.1 M oxalic acid, filtered rapidly at  $0^{\circ}$ C, the solution diluted with 1/6 its volume of ice-cold 95% alcohol and placed in a cooling mixture for crystallization. It is best to prepare the salt in a cold room, otherwise proper cooling becomes cumbersome.

#### PROPERTIES:

Green, crystalline salt; not homogeneous; consists of green and orange rods (probably the cis and trans forms). Rapidly decomposed at room temperature, particularly when exposed to light; remains stable for a few weeks at  $-6^{\circ}$ C if light is excluded. Solutions are initially green, but rapidly turn brown and become clouded; solutions containing some oxalic acid kept at 0°C remain clear for some time.

REFERENCE:

G. H. Cartledge and W. P. Ericks. J. Amer. Chem. Soc. <u>58</u>, 2061 (1936).

### Potassium Hexacyanomanganate (I)

#### K<sub>5</sub>Mn(CN)<sub>6</sub>

According to Manchot and Gall, the salt is best prepared by starting with  $Na_4Mn(CN)_6 \cdot xH_2O$  (see p. 1473). Thus, 10 g. of the

salt is dissolved in 150 ml. of 2% sodium hydroxide (Erlenmeyer flask), the air being kept out during this operation by a stream of  $H_2$ . Then, 8 g. of Al granules is added little by little, but rather rapidly (2 minutes); the sparingly soluble  $Na_2Mn[Mn(CN)_6]$  should not precipitate in the process. After about five minutes the solution becomes intensely yellow-brown; it is then rapidly suction-filtered through a Pyrex filter of small pore size. The filtrate is allowed to flow into 150 ml. of a solution containing 15 g. of KOH and 30 g. of KCN and saturated with KCl. The desired compound is thereby precipitated as a white, crystalline powder, only sparingly soluble in water; any  $Mn^{++}$  present remains in solution. The salt is separated by rapid filtration through a Pyrex filter and thoroughly washed, first with 200 ml. of 10% potassium hydroxide, then with 100 ml. of 20% KCN solution, and finally with about 700 ml. of boiled, ice-cold water (until the filtrate is completely colorless).

### PROPERTIES:

The potassium salt obtained in the above manner is slowly oxidized in moist air, turning brown. The sodium salt solution is rapidly discolored in air;  $H_2$  is liberated on boiling but there is a slow evolution even at room temperature.

**REFERENCES:** 

W. Manchot and H. Gall. Ber. dtsch. chem. Ges. <u>61</u>, 1135 (1928). Preparation by means of a sodium amalgam is described by W. D. Treadwell and W. E. Raths. Helv. Chim. Acta <u>35</u>, 2277 (1952).

### Potassium Hexacyanomanganate (II)

### $K_4Mn(CN)_6 \cdot 3 H_2O$

 $\begin{array}{rrrr} MnCO_3 &+ & 6 \ KCN &+ & 3 \ H_2O &= & K_4Mn(CN)_6 \cdot 3 \ H_2O &+ & K_2CO_3 \\ 114.9 & & 390.6 & & 54.1 & & 421.4 & & 138.2 \end{array}$ 

A paste of 20 g. of freshly precipitated  $MnCO_3$  (see the next preparation) is placed in a flask from which the air has been displaced with  $N_2$  and heated on a water bath to 70-80°C. A solution of 80 g. of KCN in 100 ml. of water is added slowly in drops, and the mixture is maintained at this temperature for an additional half hour (occasional swirling). The small residue of undissolved  $MnCO_3$  is removed by rapid filtration of the hot solution, air being excluded as completely as possible. The bluish-purple crystals which precipitate from the yellow solution on cooling are separated by suction filtration, washed with alcohol, and dried in a  $N_2$  stream at room temperature.

Alternate method: Readily prepared from  $Mn(CH_3CO)_2$  and KCN. The sodium compound  $Na_4Mn(CN)_6$  is prepared in a completely similar manner; it is more soluble than the potassium salt.

SYNONYM:

Potassium manganese (II) cyanide.

PROPERTIES:

Soluble without being altered only in solutions which have a KCN concentration higher than 1.5 N; at lesser CN concentrations, the greenish  $K_2Mn[Mn(CN)_6]$  is precipitated. The crystals effloresce in air with partial oxidation.

**REFERENCES:** 

G. Grube and W. Brause. Ber. dtsch. Chem. Ges. <u>60</u>, 2273 (1927);
 J. Meyer. Z. anorg. allg. Chem. <u>81</u>, 390 (1913); P. Straus.
 Z. anorg. Chem. <u>9</u>, 6 (1895).

### Potassium Hexacyanomanganate (III)

### K<sub>3</sub>Mn(CN)<sub>6</sub>

MnSO <sub>4</sub> ·	H₂C	) + 2N	аH	CO3 =	= M	InCO3	$+ N_{2}$	a2SO4	$+ CO_2$	+	$2\mathrm{H_2O}$
1 <b>69.0</b>		10	<b>3</b> 8.0		1	14 <b>.9</b>					
2 MnCO <sub>3</sub>	+	12 KCN	+	H <sub>2</sub> O <sub>2</sub>	=	2 K₃M	In(CN)	), +	2 K <sub>2</sub> CO <sub>3</sub>	+	2 KOH
<b>22</b> 9.9		781 <b>.2</b>		34.02		6	5 <b>6.</b> 5		276.4		112.2

A fresh precipitate of  $MnCO_3$  is prepared by slowly adding a solution of 50 g. of  $MnSO_4 \cdot H_2O$  in 120 ml. water to a solution of 75 g. of NaHCO<sub>3</sub> in 950 ml. of water (20°C, good stirring; caution: the mixture tends to foam). The product is filtered off with suction, washed with a large amount of water, and, while still wet, thoroughly mixed with a solution of 135 g. of KCN in 270 ml. of water, producing a dark-blue solution of K<sub>4</sub>Mn(CN)<sub>6</sub>. Following cooling to ~15°C, 150 ml. of 3% H<sub>2</sub>O<sub>2</sub> is added slowly and with sitrring. The solution is allowed to stand for a few minutes until its color turns deep dark brown. With sufficient cooling no appreciable amount of O<sub>2</sub> is evolved.

The solution is now passed without delay through a suction filter (to remove any small residues of  $MnO_2$  and similar compounds) and allowed to crystallize overnight in a refrigerator. The precipitated crystals (60-70 g., 63-74% yield based on MnSO<sub>4</sub>)

1474

are filtered off with suction, washed with alcohol and dried in a desiccator. An additional crop can be recovered from the mother liquor by covering it carefully with a layer (roughly the same volume) of alcohol and allowing to stand for several days.

If it is necessary to recrystallize the salt, it is covered with 8-10 times its volume of 10% KCN, rapidly heated on a water bath to  $45^{\circ}$ C (stirring), immediately suction-filtered, cooled with ice and covered with alcohol as described above.

SYNONYM:

Potassium manganese (III) cyanide.

PROPERTIES:

Dark red-brown needles; stable in air; decomposed by water, forming hydrated  $Mn_2O_3$ .

REFERENCES:

This procedure was developed in (unpublished) experiments in cooperation with E. Brodkorb; G. Grube and W. Brause. Ber. dtsch. chem. Ges. <u>60</u>, 2273 (1927); J. Meyer. Z. anorg. allg. Chem. <u>81</u>, 390 (1913).

## SECTION 26

Rhenium

O. GLEMSER

## Rhenium Metal

Prepared by reduction of NH<sub>4</sub>ReO<sub>4</sub> or KReO<sub>4</sub> with H<sub>2</sub>.

I. Fine  $NH_4ReO_4$  powder is slowly heated to 200-250°C in very pure  $H_2$  and held at that temperature for three hours. The temperature is then raised to 500°C and the reduction completed at 1050°C (six hours). The boats and reactor tubes should be porcelain.

If heating rates are too high, part of the product evaporates as the oxide and deposits to form a mirror on the cooler parts of the tube.

II. KReO<sub>4</sub>, in a silver boat, is reduced in very pure H<sub>2</sub> at 500°C. The product is extracted with H<sub>2</sub>O containing a small amount of HCl, dried and again reduced with H<sub>2</sub> in a porcelain boat at 1000°C.

PROPERTIES:

Atomic weight 186.22. Gray metal powder; the solid has a platinum-like luster. M.p. 3170°C; d 20.35; Brinell hardness 250. Readily soluble in nitric acid and slowly in sulfuric acid. Crystal structure: A 3 type.

REFERENCES:

- I. W. Biltz and G. A. Lehrer. Nachr. Gött. Ges. 1931, 193.
- II. W. Biltz. Z. Elektrochem. <u>37</u>, 498 (1931); W. Geilmann. Private communication.

### Rhenium (III) Chloride

### **ReCl**<sub>3</sub>

 $\begin{array}{rcl} \mathrm{Re} & + & \frac{3}{2} & \mathbf{Cl}_{2} & = & \mathrm{Re}\mathbf{Cl}_{3} \\ \mathrm{186.2} & & \mathrm{33.0} \ l. & & \mathrm{292.6} \end{array}$ 

Rhenium metal is placed in a reactor consisting of a hard glass tube joined to a receiver manifold with seven bulbs RHENIUM

sealed on. The air is displaced with oxygen-free  $H_2$  and the Re then heated in a stream of  $Cl_2$ . The raw sublimate is collected in the first bulb; it is resublimed into the second bulb under oxygen-free, dry  $N_2$  (the less volatile ReCl<sub>3</sub> remains in the first bulb). The operation is repeated using the next set of bulbs, etc. The ReCl<sub>3</sub> fractions are then collected from all the bulbs and resublimed at 2-3 mm. and 500-550°C.

ANALYSIS:

Oxidation to  $\text{ReO}_4^-$  with sodium hydroxide +  $H_2O_2$ ; the  $\text{ReO}_4^-$  ion is precipitated as nitron hydrogen perrhenate.

PROPERTIES:

Dark purple-red crystals. Bimolecular under normal conditions (Re<sub>2</sub>Cl<sub>6</sub>). Converted in moist air to ReCl·  $2H_2O$  (2-3 hours); the water of hydration is readily removed by heating to 100°C in vacuum over  $P_2O_5$ . Soluble in water with a deep dark-red color; the solution turns cloudy after several hours because of hydrolysis, and black  $Re_2O_3 \cdot H_2O$  is precipitated. Complete hydrolysis on boiling. Soluble in glacial acetic acid and dioxane (reddish-purple color), alcohol and liquid ammonia; slightly soluble in ether. A AgNO<sub>3</sub> solution produces a precipitate only after lengthy heating with nitric acid. Forms well-crystallized compounds with RbCl, CsCl and organic bases. Hexagonal crystal structure.

REFERENCES:

W. Geilmann, F. W. Wrigge and W. Biltz. Nachr. Gött. Ges. <u>1932</u>, 582; W. Geilmann and F. W. Wrigge. Z. anorg. allg. Chem. <u>214</u>, 249 (1933); O. W. Kolling. Trans. Kansas Acad. Sci. <u>56</u>, 378 (1953).

## Rhenium (V) Chloride

#### **ReCl**<sub>5</sub>

 $\begin{array}{rcl} \operatorname{Re} &+& {}^{5}\!/_{2}\operatorname{Cl}_{2} &=& \operatorname{Re}\operatorname{Cl}_{5} \\ 186.2 & 55.0 \ l. & 363.5 \end{array}$ 

Rhenium metal is placed in a boat which is inserted into the hard glass apparatus of Fig. 328. The air is displaced by  $O_2$ -free nitrogen and the Re chlorinated at 500°C in a stream of  $Cl_2$ . The evolving black-brown vapors are condensed at a as a black solid. The apparatus is sealed off at 1, connected at 5 to a high vacuum,



Fig. 328. Preparation of rhenium (V) chloride

evacuated and heated from 20 to  $50^{\circ}$ C; small fractions of the very volatile ReOCl<sub>4</sub> (b. p. 223°C) are then condensed in *d* and *e*. The bulk of the ReCl<sub>5</sub> is driven into *b* at 150 to 250°C, the tube melt-sealed at 2, and the substance sublimed at 200°C from *b* to *c*, leaving only a slight residue in *b*. Finally, the tube is melt-sealed at 3 and 4 and the preparation distributed (by shaking) into the small tubes attached at *c*; the latter are then melt-sealed.

PROPERTIES:

A deep, black-brown powder; dark brown vapor. Sensitive to air, sublimation at atmospheric pressure results in decomposition. Hydrolyzed by water, forming various products. Soluble in hydrochloric acid (green solution) with liberation of  $Cl_2$ .

**REFERENCES:** 

W. Geilmann, F. W. Wrigge and W. Blitz. Angew. Chem. <u>46</u>, 223 (1933); Z. anorg. allg. Chem. <u>212</u>, 244 (1933).

## Potassium Rhenium (IV) Chloride

### K<sub>2</sub>ReCl<sub>6</sub>

Fine Re powder is intimately ground with KCl (10% excess) and slowly heated to about 300°C in a porcelain boat, first under  $N_2$  and then in a slow  $Cl_2$  stream. The  $K_2ReCl_6$  is formed immediately and only a slight quantity of rhenium chlorides is volatilized. Following cooling in a stream of  $N_2$ , the substance is dissolved in some hot 5% HCl and recrystallized; the remainder is obtained by concentration and crystallization during cooling.

Alternate method: Reaction of KReO<sub>4</sub> with KI and hydrochloric acid. The procedure is involved and it is difficult to obtain a pure product [H. Schmidt, Z. anorg. allg. Chem. <u>212</u>, 188 (1933); O. W. Kolling, Trans. Kansas Acad. Sci. <u>56</u>, 379 (1953)].

PROPERTIES:

Yellowish-green powder or regular green crystals. Melts with decomposition. Addition of conc.  $H_2SO_4$  at moderate temperatures produces HCl. Fair solubility in water. Solubility in 12% HCl: 21.4 (0°C); 30.3 (18°C) g./liter; in 37% HCl: 3.3 (0°C); 3.7 (18°C) g./liter.  $d_4^{15}$  3.34. Crystal structure: type J1<sub>1</sub>.

REFERENCE:

W. Geilmann. Private communication.

### Rhenium (VI) Oxychloride

### ReOCl<sub>4</sub>

Prepared via reaction of Re<sub>2</sub>O<sub>7</sub> with ReCl<sub>5</sub>.

Rhenium metal is chlorinated at 500°C in a stream of  $Cl_2$  in the apparatus described for the preparation of  $ReO_3Cl$ . Following cooling, the  $Cl_2$  is displaced with  $O_2$  and the sections of the tube in which brown-black crystals of  $ReCl_5$  have appeared are heated with a small flame to 50-70°C. The  $ReCl_5$  melts (often with appearance of a flame) and the crystals turn into a brown liquid, which is then distilled in a stream of  $N_2$  into a well-cooled U tube receiver. The excess  $Cl_2$  is evaporated, the apparatus filled with  $O_2$ , and the liquid brought to a gentle boil. Heating at 200°C in an  $N_2$  stream is continued for an hour in order to completely remove all traces of  $ReO_3Cl$ . Then about one third of the remaining liquid is distilled off. The receiver is now replaced by a fresh one and, except for a small residue, the remaining liquid is distilled over.

Alternate method: From ReCl<sub>3</sub> and dry  $O_2$  at 110 to 130°C [O. W. Kolling, Trans. Kansas Acad. Sci. <u>56</u>, 378 (1953)].

PROPERTIES:

Formula weight 344.05. Fibrous needles; dark orange in thin, brownish-red in thick layers. M. p. 29.3 °C, b. p. 223 °C (slight decomposition). Decomposes at 300 °C. Immediately forms  $\text{ReO}_3$  Cl on heating in a stream of O<sub>2</sub>. Hydrolyzed by water to rhenium (IV) hydroxide and HReO<sub>4</sub>.

REFERENCE:

A. Brukl and K. Ziegler. Ber. dtsch. chem. Ges. <u>65</u>, 916 (1932).

## Rhenium (VII) Oxychloride

## ReO<sub>3</sub>Cl

Prepared by reaction of  $\operatorname{Re}_2O_7$  (excess) with  $\operatorname{ReCl}_5$ .

Two boats, one containing five and the other two parts of Re, are placed in a high-melting glass reactor tube in a manner such that heating of one will not raise the temperature of the other. A U-shaped tube and two condensation traps are connected to the reactor by means of ground-glass joints (the traps are cooled to -65°C with alcohol-Dry Ice). The air in the apparatus is displaced with O<sub>2</sub> and the first boat (the one containing five parts of Re) is heated in a slow stream of O<sub>2</sub> in such a way that the  $\operatorname{Re}_{2}O_{7}$  formed is deposited in the tube (the U tube must be cooled to a low temperature during this operation). The oxygen is then displaced with Cl<sub>2</sub> and the second boat heated in a stream of Cl<sub>2</sub>. The rhenium chlorides formed in this manner react with the  $\text{Re}_{2}O_{7}$ and the products of this reaction are condensed in the U tube. Any cocondensed  $Cl_2$  is evaporated; then, the  $ReO_3Cl$  is distilled over as the first fraction boiling above 100°C (it is usually very light blue or green). On repeated fractionation in a stream of N<sub>2</sub> the product becomes colorless.

Alternate method: From ReO<sub>3</sub> and dry Cl<sub>2</sub> at 160-190°C. Yields exceed 70% [C. J. Wolf, A. F. Clifford and W. H. Johnston, J. Amer. Chem. Soc. <u>79</u>, 4257 (1957)].

PROPERTIES:

Formula weight 269.68. Colorless liquid: strongly light refracting. M.p. 4.5°C, b.p. 131°C (corr.). Reacts instantaneously with Hg, Ag, stopcock grease and numerous other organic compounds. Soluble in CCl<sub>4</sub>. Hydrolyzes to HReO<sub>4</sub> and HCl.

REFERENCE:

I.

A. Brukl and K. Ziegler. Ber. dtsch. chem. Ges. 65, 916 (1932).

## Rhenium (IV) Oxide

### ReO<sub>2</sub>

$$2 \operatorname{Re}_{2}O_{7} + 3 \operatorname{Re} = 7 \operatorname{Re}O_{2}$$
  
 $1_{1_{12}}$ : 96.9 55.9 152.8

A stoichiometric mixture of Re and  $\text{Re}_{2}O_{7}$  is heated to 300°C for one day in a small, evacuated, thick-wall quartz tube, which is

#### RHENIUM

sealed by melting; the reactants are then heated to 600-650°C for an additional day. The product is orthorhombic  $\text{ReO}_{\approx}$ .

II. Heating of  $NH_4ReO_4$  in vacuum at 500°C yields monoclinic  $ReO_2$  of the  $MoO_2$  type; above 500°C, orthorhombic oxides are formed.

PROPERTIES:

Formula weight 218.22. Gray-black powder; dissociates in a high vacuum at 1000°C to Re and Re<sub>2</sub>O<sub>7</sub>. Readily oxidized by O<sub>2</sub>. Insoluble in weak acids, but dissolved by conc. halogen acids. Converted to HReO<sub>4</sub> by H<sub>2</sub>O<sub>2</sub> and HNO<sub>3</sub>.  $d_4^{25}$  11.4. Heat of formation: -70 kcal.

**REFERENCES**:

- I. W. Biltz. Z. anorg. allg. Chem. 214, 227 (1933).
- II. W. H. Zachariasen. Amer. Crystallographic Assoc. Program and Abstracts of Winter Meeting, F4 (1951); A. Magnéli. Acta Crystallogr. (Copenhagen) 9, 1038 (1956).

## Rhenium (VI) Oxide

#### ReO<sub>3</sub>

I. REDUCTION OF Re<sub>2</sub>O<sub>7</sub> BY CARBON MONOXIDE

 $\begin{array}{rcl} \operatorname{Re}_{2}O_{7} \ + \ \operatorname{CO} \ = \ 2 \ \operatorname{Re}O_{3} \ + \ \operatorname{CO}_{2} \\ 484.4 \ & 22.4 \ l. \ & 468.4 \ & 22.4 \ l. \end{array}$ 

The apparatus is a glass tube sealed at one end; about 1 g. of  $\operatorname{Re}_2O_7$  is sublimed into it in a stream of  $O_2$ . When the reaction is completed, the apparatus is evacuated and filled with CO to a pressure of 760 mm. The glass tube is then slowly heated to 175°C in a glycerol bath and held at that temperature until the preparation turns blue. The temperature is then slowly raised to 225°C and later, when red  $\operatorname{ReO}_3$  is formed, to 280°C. The run requires two to three hours. The yield is quantitative.

II. Reaction of  $\text{Re}_3O_7$  with dioxane to form a complex compound, which dissociates at 125-145°C to  $\text{ReO}_3$  and some volatile products.

The apparatus consists of a reaction flask protected against moisture; 4 ml. of dioxane is rapidly added to 1 g. of  $\text{Re}_2O_7$  under anhydrous conditions (the dioxane should be predistilled

over Na metal). The mixture is gently heated on a water bath until a clear, colorless solution is obtained. Local overheating must be avoided, since it produces cloudy solutions and, ultimately, contaminated products. The flask is then placed in an ice bath to freeze the solution. After the freezing, the frozen substance allowed to melt. The  $\operatorname{Re}_2O_7$ -dioxane complex crystallizes is in the form of a dense, pearly-gray precipitate; the excess dioxane becomes liquid. The freezing-melting operation is then repeated, the excess of dioxane decanted, and the compound dried in a vacuum desiccator at room temperature over conc. H<sub>2</sub>SO<sub>4</sub>. The dry complex is rapidly placed in a crucible and carefully heated on a hot plate (125 to 145°C). The substance is melted, forming a colorless to bluish-green liquid, which later dissociates to red ReO<sub>3</sub> and some volatile, Re-free products. The ReO<sub>3</sub> thus formed is pure. The yield is about 95%.

### PROPERTIES:

Formula weight 234.22. Red powder. During reduction of  $\text{Re}_2\text{O}_7$ , the preparation passes through intermediate stages with hues ranging from green to dark blue (rhenium blue) until the final red color is obtained. Decomposed in high vacuum at 400°C to  $\text{Re}_2\text{O}_7$  and  $\text{ReO}_3$ . Not attacked by hot hydrochloric acid, but converted to  $\text{HReO}_4$  by strong  $\text{HNO}_3$ . Disproportionates in warm NaOH to  $\text{ReO}_2$  and  $\text{NaReO}_4$ ; NaOH + H<sub>2</sub>O<sub>2</sub> instantaneously produce  $\text{NaReO}_4$ .  $d_4^{25}$  6.9; heat of formation: --146.0 kcal. Crystal structure:  $\text{DO}_9$  type.

### REFERENCES:

- I. A. D. Melaven, J. N. Fowle, W. Brickel and C. F. Hiskey in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York-Toronto-London, 1950, p. 187.
- II. H. Nechamkin and C. F. Hiskey. Ibid., p. 186; H. Nechamkin, A. N. Kurtz, and C. F. Hiskey. J. Amer. Chem. Soc. <u>73</u>, 2828 (1951).

## Rhenium (VII) Oxide

### Re<sub>2</sub>O<sub>7</sub>

$$2 \operatorname{Re} + \frac{7}{2} \operatorname{O}_{2} = \operatorname{Re}_{2} \operatorname{O}_{7}$$
  
372.4 78.4 l. 484.4

The compound is prepared in a combustion tube to which a U tube is sealed on; the latter is attached to a condensation trap.

1482

#### RHENIUM

Both ends of the tube are protected against humidity by vessels containing  $CaCl_2$  and conc.  $H_2SO_4$ . The rhenium metal is placed in a porcelain boat situated in the front section of the tube, which is heated to  $150^{\circ}C$  while a very fast stream of oxygen is passed over the metal. Crystals of  $Re_3O_7$  deposit in the front section of the tube and in the condenser tube (there is virtually no vapor mist); if the starting Re is not entirely alkali-free, some KReO<sub>4</sub> will remain in the boat. The  $Re_2O_7$  should be resublimed in a stream of  $O_2$ .

The compound is used in the preparation of HReO  $_{4}$ , of the lower oxides and of very pure Re (reduction with H<sub>2</sub>).

#### PROPERTIES:

Bright-yellow crystalline powder. M. p.  $301.5^{\circ}$ C, b.p.  $362.4^{\circ}$ C;  $d_4^{25}$  6.103; heat of formation:-295.9 kcal. Extremely hygroscopic. Readily soluble in H<sub>2</sub>O, forming HReO<sub>4</sub>; soluble in alcohol; sparingly soluble in ether. Stored in melt-sealed glass tubes. Rhombic crystals.

REFERENCES:

W. Geilmann. Private communication; W. A. Roth and C. Becker.
 Z. phys. Chem. <u>159</u>, 29 (1932); K. Wilhelmi. Acta Chem. Scand.
 8, 693 (1954).

### Sodium Rhenate (IV)

### Na<sub>2</sub>ReO<sub>3</sub>

 $\begin{array}{rrrr} {\rm ReO_2} &+& 2 \ {\rm NaOH} &=& {\rm Na_2ReO_3} &+& {\rm H_2O} \\ {\rm 218.2} && {\rm 80.0} && {\rm 280.2} \end{array}$ 

The apparatus is shown in Fig. 329. A gold crucible c is introduced into the quartz apparatus through opening e and placed in



Fig. 329. Preparation of sodium rhenate (IV). a quartz tube; b crucible-shaped attachment; c gold crucible; d thermocouple; l movable gold shield (splash shield).

the attachment b; a stream of oxygen-free N<sub>2</sub> is passed from h to g. After thorough flushing with N<sub>2</sub>, 10 g. of very pure and dry NaOH is introduced (through f) into c. The apparatus is then heated to 500 °C and 4 g. of dry ReO<sub>2</sub> added by shaking to the melt. The ReO<sub>2</sub> dissolves, the melt turns reddish-brown, and water vapor escapes through f. After the reaction is completed, the heater is removed and attachment b cooled with ice water; this spalls the solidified melt off the crucible walls. The fused material contains (clearly separated) an upper layer of NaOH and a lower stratum of the brown rhenate. The fused cake is gently crushed, leached with deaerated ice water, decanted, filtered in the absence of air, washed with alcohol and dried, again in the absence of air. The filtering and drying device shown in Fig. 52, p. 74, is handy in this operation.

Alternate method: Fusion of ReO<sub>3</sub> with NaOH [W. Geilmann, F. W. Wrigge and W. Biltz, Z. anorg. allg. Chem. <u>214</u>, 233 (1933)].

SYNONYM:

Sodium rhenite.

PROPERTIES:

Brown powder. On heating in air, converts to the yellow perrhenate. Insoluble in  $H_2O$  and in bases; soluble in conc. hydrochloric acid, converting to the green  $H_2ReCl_6$ .

REFERENCE:

I. and W. Noddack. Z. anorg. allg. Chem. 215, 134 (1933).

## Ammonium Perrhenate

## NH<sub>4</sub>ReO<sub>4</sub>

 $\frac{\text{Re}_2\text{O}_7 + 2 \text{ NH}_3 + \text{H}_2\text{O}}{484.4} = 2 \frac{\text{NH}_4\text{ReO}_4}{536.6}$ 

A solution of  $\text{Re}_2O_7$  in some water is prepared, excess ammonia is added and the solution evaporated in a platinum crucible placed on a water bath.

Alternate method: Rhenium sulfides, oxides or rhenium metal are dissolved in  $HNO_3$ . The solution is evaporated and diluted with ammonia. Recrystallization is required! May be used to prepare pure rhenium (see p. 1476).

1484

PROPERTIES:

Thick, white, hexagonal crystals. Dissociates in air above 200°C to form NH<sub>3</sub>, H<sub>2</sub>O and Re<sub>2</sub>O<sub>7</sub>. Solubility: 2.9 (0°C); 6.2 (20°C); 32 (80°C) g. of salt/100 g. H<sub>2</sub>O; d 3.63. Crystal structure: H0<sub>4</sub> type.

REFERENCE:

I. and W. Noddack. Z. anorg. allg. Chem. <u>181</u>, 23 (1929).

### Barium Perrhenate

### Ba(ReO<sub>4</sub>)<sub>2</sub>

An aqueous solution of  $\text{Re}_2O_7$  (=  $\text{HReO}_4$ ) is neutralized exactly with baryta water, using neutral red as the indicator. The residue is dried, and the water-containing salt is converted to anhydrous Ba( $\text{ReO}_4$ )<sub>2</sub> in vacuum or by heating to 120°C.

PROPERTIES:

Formula weight 637.80. Colorless columns or rhomboids. Solubility: 1.8 (0°C); 5.3 (20°C); 47 (70°C) g. of salt/100 g.  $H_2O$ . Solubility in alcohol: 2.4 g/liter of solution at 18.5°C.

**REFERENCES:** 

I. and W. Noddack. Z. anorg. allg. Chem. <u>181</u>, 25 (1929); W. Lewino. Thesis, Univ. of Hamburg, 1932.

### Barium Rhenate (VI)

### BaReO<sub>4</sub>

Prepared by reducing Ba(ReO<sub>4</sub>)<sub>2</sub> with ReO<sub>2</sub> and NaOH in a melt. The apparatus is the same as that used in the preparation of Na<sub>2</sub>ReO<sub>3</sub> (Fig. 329). Sodium hydroxide (20 g.) is fused in crucible c under a stream of N<sub>2</sub>; then, 8.00 g. of Ba(ReO<sub>4</sub>)<sub>2</sub> is added, followed by 2.00 g. of ReO<sub>2</sub>. The melt is heated to 500°C and held at that temperature for one hour. It is then cooled to 300°C and held at that temperature for one hour. The heater is then removed and attachment b cooled with ice water. The melt cake is broken up and treated with 96% alcohol at 0°C; this loosens the NaOH and NaReO<sub>4</sub> in the melt: the former reacts with any unconverted Ba(ReO<sub>4</sub>)<sub>2</sub> to form NaReO<sub>4</sub> and Ba(OH)<sub>2</sub>; the latter remains in the residue. The product is filtered, using the device shown in Fig. 52, p. 74, and washed with some alcohol. It contains some NaOH and Ba(OH)<sub>2</sub>.

ANALYSIS:

The compound is disproportionated in acetic acid. The leachedout Re (VII) fraction is filtered off and precipitated with nitron; the Re (IV) fraction is oxidized to Re (VII) and also precipitated with nitron.

### PROPERTIES:

Formula weight 387.58. Foliage-green powder; readily dissociated. Slowly turns black in vacuum and white in air (formation of  $\text{ReO}_{4}$ ). Instantaneously dissociated by water, acids and bases. The presence of a slight amount of free NaOH is required for stability.

REFERENCE:

I. and W. Noddack. Z. anorg. allg. Chem. 215, 143 (1933).

186.2

## Rhenium (IV) Sulfide

ReS<sub>2</sub>

 $Re + 2S = ReS_2$ 

64.1

250.3

I.

Rhenium and sulfur are mixed in stoichiometric proportions and heated for 18 hours in an evacuated, sealed small quartz tube at 980-1000°C.

II. Hydrogen sulfide is used to precipitate  $\operatorname{Re}_2 S_7$  from a hydrochloric acid solution; the precipitate is filtered off with suction and washed with water and briefly with acetone. It is dried in a quartz or hard glass tube sealed at one end. It is then heated in high vacuum to 600°C until no further S sublimes out. The result is  $\operatorname{ReS}_2$  only slightly contaminated with sulfur.

Alternate method: Heating of Re in a stream of  $H_2S$  to red heat [H. V. A. Briscoe, P. C. Robinson and E. M. Stoddart, J. Chem. Soc. (London) <u>1931</u>, 1441].

PROPERTIES:

Black solid (platelets are seldom recognizable); somewhat volatile at 1000°C. Strongly attacks quartz at 1000°C. No appreciable solubility in bases, alkaline sulfides, hydrochloric and sulfuric

RHENIUM

acids. Converted by oxidizing agents to HReO<sub>4</sub>.  $d_{2}^{20}$  7.506; heat of formation: -70.5 kcal. Crystal structure: C 7 type.

REFERENCES:

I. R. Juza and W. Biltz. Z. Elektrochem. <u>37</u>, 499 (1931). II. W. Geilmann and G. Lange. Z. analyt. Chem. 126, 321 (1953).

### Rhenium (VII) Sulfide

#### $Re_2S_7$

 $2 \text{ KReO}_4 + 7 \text{ H}_2\text{S} + 2 \text{ HCl} = \text{Re}_2\text{S}_7 + 2 \text{ KCl} + 8 \text{ H}_2\text{O}$ 578.6 155.0 l. 72.9 596.9 149.1

A solution of KReO<sub>4</sub>, containing 30 ml. of hydrochloric acid per 100 ml. of solution, is saturated for four hours with H<sub>2</sub>S. The precipitated sulfide is washed with H<sub>2</sub>S-saturated, 3% HCl water. The product is filtered in the absence of air, washed and then dried, first in a high vacuum for two hours at 140°C and then in high vacuum over freshly prepared  $P_2O_5$  (60 hours at 165-170°C).

Alternate method: Precipitation with compressed  $H_2S$  from a solution of KReO<sub>4</sub> in hydrochloric acid; there is no need for the high HCl concentration in this case. The workup is similar to that described above [W. Geilmann, Z. analyt. Chem. <u>126</u>, 321 (1943)].

**PROPERTIES:** 

Black, readily oxidized powder. Dissociation to  $\text{ReS}_2$  and S begins at 250°C. Insoluble in hydrochloric and sulfuric acids in the absence of air; oxidized by nitric acid or  $H_2O_2$  plus a base to HReO<sub>4</sub>.  $d_4^{25}$  4.866.

REFERENCE:

W. Biltz and F. Weibke. Z. anorg. allg. Chem. 203, 4 (1931).

#### Barium Mesoperrhenate

#### Ba<sub>3</sub>(ReO<sub>5</sub>)<sub>2</sub>

Prepared by fusion of Ba(ReO<sub>4</sub>)<sub>2</sub> with NaOH. The apparatus is the same as used in the preparation of Na<sub>2</sub>ReO<sub>3</sub> (Fig. 329).

Crucible c is used to fuse 3 g. of Ba(ReO<sub>4</sub>)<sub>2</sub> with 5 g. of carbonate-free NaOH under a stream of  $CO_2$ -free air. The hot melt is red and cloudy. After cooling, it is crushed, leached with 90% alcohol to remove the excess NaOH and the NaReO<sub>4</sub> formed in the process, then filtered, again washed with alcohol and dried with suction.

ANALYSIS:

The salt is decomposed with  $CO_2$ -free water; the Ba is precipitated as BaSO<sub>4</sub> and the ReO<sub>4</sub><sup>-</sup> as nitron hydrogen perrhenate.

PROPERTIES:

Formula weight 944.52. Small, lemon-yellow hexagonal tablets and columns. Turns red upon heating to 800°C, returns to yellow on cooling. Stable in dry air. The wet salt is decomposed by  $CO_2$ into BaCO<sub>3</sub> and Ba(ReO<sub>4</sub>)<sub>2</sub>.

REFERENCE:

I. and W. Noddack. Z. anorg. Chem. 215, 146 (1933).

## Workup of Rhenium Residues

Precipitates of nitron hydrogen perrhenate from analyses of rhenium are collected and stored separately from rhenium solutions.

Workup of nitron precipitates: The material is carefully decomposed in a stream of  $H_2$ , the products washed and oxidized to  $\operatorname{Re}_2O_7$ , and the latter dissolved in  $H_2O$  and concentrated in the presence of KOH or ammonia. The KReO<sub>4</sub> or NH<sub>4</sub>ReO<sub>4</sub> obtained in this manner can be used without further purification.

Workup of various solutions: To avoid unnecessary contamination of the air with Re, the solution is neutralized and concentrated (if necessary, by boiling). It is then cooled and acidified with hydrochloric acid and the Re precipitated under pressure as  $\text{Re}_2S_7$ . The  $\text{Re}_2S_7$  is washed and dissolved in KOH +  $\text{H}_2\text{O}_2$ , and the KReO<sub>4</sub> is allowed to crystallize out. If the KReO<sub>4</sub> is still not sufficiently pure, it is reduced with  $\text{H}_2$  and oxidized with  $\text{O}_2$  to  $\text{Re}_2\text{O}_7$ , and the latter is used to obtain Re or the perrhenate.

If traces of Mo must be removed, the procedure is as follows: the Re is precipitated with  $H_2S$  (under pressure) as  $\text{Re}_2S_7$ , the precipitate is dissolved in KOH +  $H_2O_2$ , and the traces of Mo

1488

are removed by extraction of the neutral solution with 8-hydroxyquinoline or chloroform. Following repeated precipitation under pressure as  $\text{Re}_2S_7$ , the product is dissolved in KOH +  $\text{H}_2\text{O}_2$  and crystallized as KReO<sub>4</sub>. The KReO<sub>4</sub> is recrystallized from hot water.

SECTION 27

Iron

H. LUX

## Metallic Iron

ELECTROLYTIC IRON

The following conditions are suitable for the preparation of very pure electrolytic iron: each liter of electrolyte contains about 800 g. of very pure  $FeCl_2 \cdot 4H_2O$  (Fe<sup>3+</sup> content less than 0.05%; sulfate-free) and 1.5-2.0 g. of AlCl<sub>3</sub> · 6 H<sub>2</sub>O or 0.1 g. of CrCl<sub>3</sub> · 6 H<sub>2</sub>O. The concentration of free HCl is 0.01-0.02 N, and the temperature is 90°C or higher. The anode is made of the purest iron possible, and is wrapped in an asbestos bag. A sheet of vanadium steel serves as the cathode. The nature of cathode pretreatment is important if the deposit of electrolytic iron is to be easily stripped off. The steel sheet is first polished to a high luster, degreased by being used as a cathode in an alkaline KCN bath, rinsed with water, and after being connected to the electrical circuit, placed in a FeCl<sub>2</sub> bath. The cathodic density is 0.65-1.0 amp./in<sup>2</sup>; if the operation is of long duration or proceeds at still higher current densities (up to 2 amp./ in.<sup>2</sup>), the electrolyte must be continuously taken out of the bath, filtered, retreated, the HCl content adjusted to the one indicated above, and recycled to the bath.

The relatively soft  $\alpha$ -iron deposited contains H<sub>2</sub> but no Al or Cr. The H<sub>2</sub> may be completely removed by baking in vacuum at 950°C. Electrolytic iron is free of C or Si, and of other metals if the electrolyte itself is pure.

PROPERTIES:

M.p. 1535°C, b.p. 2730°C; d 7.86. Crystal structure of  $\gamma$ -Fe: A 2 type.

REFERENCES:

G. A. Moore. J. Metals <u>5</u>, 1443 (1953); F. Müller. Z. Elektrochem. <u>47</u>, 135 (1941); F. Halla. Korrosion und Metallschutz <u>15</u>, 380 (1939). **REDUCED IRON** 

$$2 \operatorname{Fe}(OH)_3 + 3 \operatorname{H}_2 = 2 \operatorname{Fe} + 6 \operatorname{H}_2O$$
  
213.7 67.2 l. 111.7 108.1

It is best to start from very pure Fe(OH)<sub>3</sub>; this is prepared by adding a Fe(NO<sub>3</sub>)<sub>3</sub> solution to aqueous NH<sub>3</sub> and drying the precipitate at 65°C. The product, finely ground, is placed in an aluminum or Pt boat (with Pt foil insert) and reduced in a stream of H<sub>2</sub> as the temperature is slowly raised. If the reduction temperature is lower than 550°C, the iron product is pyrophoric. As a rule, the temperature is raised slowly from 400 to 700°C (with 20 g. of Fe<sub>2</sub>O<sub>3</sub>, this requires about 40 min.) and then held constant (about 20 min.) until further H<sub>2</sub>O is produced. If less reactive starting materials (e.g., Fe<sub>2</sub>O<sub>3</sub> prepared from the mitrate) are used, it may be necessary to heat to higher temperatures (1050-1100°C) and for much longer times (60 hours or more) to insure complete reduction. The preparations are cooled in H<sub>2</sub>; their hydrogen content is minimal.

REFERENCES:

R. Fricke and L. Klenk. Z. Elektrochem. <u>41</u>, 617 (1935); R. Fricke,
O. Lohrmann and W. Wolf. Z. phys. Chem. (B) <u>37</u>, 60 (1937);
P. M. Savelevich. Trudy Inst. Chist. Khim. Reaktivov, No. 15, 51 (1937); abstract in Chem. Zentr. <u>1939</u>, I, 4583; G. P. Baxter and C. R. Hoover. Z. anorg. allg. Chem. <u>80</u>, 211 (1913).

**VERY PURE IRON**:

O. Hönigschmid, L. Birckenbach and R. Zeiss. Ber. dtsch. chem. Ges. <u>56</u>, 1473 (1923); T. W. Richards and G. P. Baxter. Z. anorg. Chem. <u>23</u>, 247 (1900); also A. Gatterer. Commentationes Pontific. Acad. Sci. <u>1</u>, 77 (1937); abstract in Chem. Zentr. <u>1938</u>, I, 1745; A. Gatterer and J. Junkes. Specola astronom. Vaticana Comun. No. 6 (1938); abstract in Chem. Zentr. <u>1938</u>, II, 2243; J. Talbot, P. Albert, M. Caron and G. Chaudron. Rev. Met. <u>50</u>, 817 (1953).

### Iron (II) Chloride

### FeCl<sub>2</sub>

The starting anhydrous  $FeCl_3$  is prepared from Fe and  $Cl_2$ . The  $Cl_2$  is then displaced by means of a stream of very dry N<sub>2</sub>, and completely dry, pure  $H_2$  is immediately introduced. The reduction proceeds rapidly at 300-350°C. It is advisable to spread the solid in a tube placed in a long electric furnace and to heat slowly, section by section, in the  $H_2$  stream. Partly unconverted FeCl<sub>3</sub> sublimes in the  $H_2$  below 300°C, while above 350°C the FeCl<sub>3</sub> tends to be reduced too far (to Fe).

The preparation of FeCl<sub>2</sub> from Fe and HCl is less satisfactory because of the higher temperatures required.

SYNONYMS:

Ferrous chloride, iron dichloride.

PROPERTIES:

White hygroscopic powder; can be resublimed in a stream of HCl at about 700°C. M.p. 674°C, b.p. 1023°C; d(25°) 3.162. Vapor pressure at 700°C: 12 mm. Readily soluble in water and alcohol. Crystal structure: C19 type.

REFERENCES:

H. Wolfram. Thesis, Techn. Hochschule Dresden, 1913; W. Kangro and E. Petersen. Z. anorg. allg. Chem. <u>261</u>, 157 (1950).

## Iron (III) Chloride

## FeCl<sub>3</sub>

 $2 \text{ Fe} + 3 \text{ Cl}_2 = 2 \text{ FeCl}_3$ 111.7 67.2 l. 324.4

A stream of  $Cl_2$  is very thoroughly dried over conc.  $H_2SO_4$ and  $P_2O_5$ , and is then liquefied by passage through a U tube cooled with Dry Ice-acetone mixture to about  $-40^{\circ}C$ . The bath temperature is then raised to  $-34.1^{\circ}C$ ; pure  $Cl_2$  volatilizes out. It passes into a very dry Pyrex tube containing the purest possible iron wire (about 0.2 mm. in diameter). The reaction takes place at 250-400°C; an excess of  $Cl_2$  should always be present and should always bubble out from the  $H_2SO_4$ -containing safety valve which terminates the reactor train. To avoid plugging the tube, the electric furnace (or the aluminum heating block) is occasionally shifted, so that a fresh condensation zone may be created.

At the end of the reaction, the preparation should be resublimed in a stream of  $Cl_2$  at about 220°C (the temperature should not exceed 300°C). All of the  $Cl_2$  is then displaced from the apparatus with very dry  $N_2$  (or air), and the product is transferred (under  $N_2$ ) to storage vessels, which are then tightly sealed.

SYNONYMS:

Ferric chloride; iron trichloride.

PROPERTIES:

Formula weight 162.2. Leaflets with a somewhat greenish metallic luster. Extremely hygroscopic. M. p. (in  $Cl_2$ ) 308°C, b. p. (calcd.) 316°C; d (25°) 2.898. Decomposes partially on sublimation in high vacuum. In the range of 160-210°C the decomposition pressure of  $Cl_2$  over solid FeCl<sub>3</sub> and FeCl<sub>2</sub> obeys the equation: log P = 11.33-5.67  $\cdot$  10<sup>3</sup>/T. Very readily soluble in water, ethyl alcohol, ethyl ether and acetone. FeCl<sub>3</sub> and FeCl<sub>2</sub> form a eutectic, m. p. 297.5°C, containing 13.4 mole % of FeCl<sub>2</sub>.

**REFERENCES:** 

G. G. Maier. Techn. Pap. Bur. Mines Washington No. 360, 40 (1925); O. Hönigschmid, L. Birckenbach and R. Zeiss. Ber. dtsch. chem. Ges. <u>56</u>, 1476 (1923); H. Schäfer. Angew. Chem. <u>64</u>, 111 (1952); for the preparation of larger amounts, see B. R. Tarr in: L. F. Audrieth, Inorg. Syntheses, Vol. III, New York, 1950, p. 191.

### Iron (II) Bromide

#### FeBr<sub>2</sub>

FeBr<sub>2</sub>, ANHYDROUS

I.

 $Fe + 2 HBr = FeBr_2 + H_2$ 55.9 161.8 215.7 22.4 l.

Very pure Fe (reduced with  $H_2$ ) is placed in an unglazed procelain boat situated in a porcelain tube and heated to about 800°C in a completely dry stream of HBr-saturated nitrogen, so that the nascent FeBr<sub>2</sub> distills out at once. The preparation is transferred (in dry  $N_2$ ) to well-sealed vessels. This last operation is facilitated by attaching a snugly fitting tube onto the exit end of the reactor; the FeBr<sub>2</sub> can then sublime into this second tube. II. Careful dehydration of FeBr<sub>2</sub>  $\cdot$  4H<sub>2</sub>O in a stream of N<sub>2</sub> and HBr.

SYNONYM:

Ferrous bromide.

PROPERTIES:

Light-yellow to dark-brown crystals; hygroscopic. M.p. 684°; d (25°) 4.624. Crystal structure; C 6 type,

REFERENCES:

- I. G. P. Baxter. Z. anorg. Chem. 38, 236 (1904).
- II. G. P. Baxter, Th. Thorvaldson and V. Gobb. Z. anorg. Chem. <u>70</u>, 333 (1911).

FeBr<sub>2</sub>, HYDRATE

$$Fe + 2 HBr + 6 H_2O = FeBr_2 \cdot 6 H_2O + H_2$$
  
55.9 161.8 323.8

To prepare the hexahydrate, pure iron is dissolved in aqueous HBr and the solution evaporated below 49°C. Above this temperature one can obtain the tetrahydrate, and above 83°C the dihydrate.

Alternate method: Shaking  $Br_2$  and water with an excess of iron powder.

PROPERTIES:

Pale-green rhombic tablets; not deliquescent.

**REFERENCE:** 

F. Schimmel. Ber. dtsch. chem. Ges. 62, 963 (1929).

# Iron (III) Bromide FeBr<sub>s</sub>

A Pyrex tube, sealed at the left end and provided with a  $45^{\circ}$  bend in the middle, is connected at the right to a high-vacuum pump and a supply of Br<sub>2</sub>. Reduced iron powder is introduced into the left leg with the aid of a long-stem funnel, and the right end of the tube is then drawn out to a capillary. The Fe is thoroughly degassed by evacuation and heating. Bromine is then condensed on

1494

the iron (liquid nitrogen bath); the  $Br_2$  excess is such that after completion of the reaction the pressure in the tube will still be at least 5 atm.

After the pump end of the reactor tube is sealed off, the  $Br_2$  is condensed in the right leg by cooling the latter in ice water. The left leg is then heated to 175-200 °C (maximum) and then the right leg to 120 °C; this produces a  $Br_2$  pressure of about 5 atm. The heating is done with two mating aluminum heating blocks. The butting ends are cut at an angle of 22.5° and separated only by a thin disk of asbestos (Fig. 330).

The reaction begins, although slowly, even at room temperature, but cannot be completed at this temperature even in several months. However, under the conditions given above, pure FeBr<sub>3</sub> condenses outside the 200°C zone. If the temperature of the Fe is too high or the Br<sub>2</sub> pressure too low, some yellow FeBr<sub>2</sub> is also deposited at the left end of the tube.



Fig. 330. Preparation of iron (III) bromide.

SYNONYM:

Ferric bromide.

PROPERTIES:

Lustrous black plates; very hygroscopic. The decomposition pressure of  $Br_2$  at 90°C is 55 mm. and at 139°C is 760 mm.; below 139°C, it obeys the equation: log p (mm.) = -3478.6/T + 11.327. The vapor pressure of FeBr<sub>3</sub> becomes detectable above 139°C. Crystal structure: D0<sub>5</sub> type (same as FeCl<sub>3</sub>).

REFERENCES:

N. W. Gregory and B. A. Thackrey. J. Amer. Chem. Soc. <u>72</u>, 3176 (1950); N. W. Gregory. Ibid. <u>73</u>, 472 (1951).

> Iron (II) lodide FeI<sub>2</sub> Fe + I<sub>2</sub> = FeI<sub>2</sub> 55.8 253.8 309.7

I.

A high-melting glass tube is sealed at one end and provided with a side arm terminating in a break-seal capillary d (see Fig. 331). Very pure iron wire is placed at location a of the tube, and a small plug of freshly ignited asbestos wool b is inserted close to the iron. Pure, dry  $I_2$  (less than stoichiometric quantity), in the form of a coarse powder, is then placed on top of the asbestos wool plug. The reactor tube is then drawn out to smaller size at a, the  $I_2$  is moved forward to c, and a good vacuum is applied by means of a mercury diffusion pump. At the same time, the other sections of the tube (a, b) are heated in an electric furnace to 500°C to degas the Fe and the asbestos. The tube is then sealed off at f, and section c containing the  $I_2$  is heated to 180°C with a suitable Al block or in an air bath (the section containing the Fe is maintained at 500°C). The tube is slightly inclined forward to prevent the liquid iodine from flowing into the hot section. If the above temperatures are adhered to, the internal pressure does not exceed one atmosphere.

The nascent FeI<sub>2</sub> sublimes slowly toward the cooler zone, where it deposits as black leaflets which appear brownish red when viewed by transmitted light. As soon as the iodine vapor disappears completely, the tube is allowed to cool and a vacuum hose is carefully slipped over side arm d. The hose is connected via a three-way stopcock to a canned-rotor pump and a nitrogen supply. Vacuum is applied, the sealed capillary on the side arm is broken off, and pure, dry N<sub>2</sub> is slowly admitted. Care should be taken during this step to avoid entraining any glass fragments in the tube. As soon as a slight gage pressure is established, the reactor tube is broken off at e and the product is transferred in nitrogen to a storage vessel.



Fig. 331. Preparation of iron (II) iodide. a iron wire; b asbestos wool; c iodine; f seal location (after sealing, the section to the right of f is removed).

Alternate methods: II. Heating reduced Fe in a stream of  $I_2$ -saturated hydrogen, followed by distillation in a steel tube.

III. Thermal decomposition of  $Fe(CO)_4I_2$ . The product is an extremely fine powder.

SYNONYM:

Ferrous iodide.

PROPERTIES:

M. p. 587°. Hygroscopic; becomes whitish in air. Aqueous solutions are colorless. Crystal structure: C 6 type.

REFERENCES:

- I. M. Guichard. Comptes Rendus Hebd. Séances Acad. Sci. <u>145</u>, 807 (1907).
- II. C. L. Jackson and I. H. Derby. Amer. Chem. J. <u>24</u>, 16 (1900); Bull. Soc. Chim. France [3] <u>24</u>, 863 (1900); see also W. Fischer and R. Gewehr. Z. anorg. allg. Chem. <u>222</u>, 303 (1935).
- III. W. Hieber and H. Lagally. Ibid. 245, 300, 313 (1940).

## Iron (II) Oxide

### FeO

I.

$$FeC_2O_4 = FeO + CO + CO_2$$
  
143.8 71.8 28.0 44.0

Thermal decomposition of  $FeC_2O_4$  yields pure FeO only under specific conditions. The decomposition is carried out in a quartz vessel (Fig. 332) whose lower section is kept at 850°C by means of an electric furnace. The joint is surrounded by a water-cooled lead coil or a rubber hose. The nascent gases should be removed as quickly as possible; for this reason, the reactor is connected to two parallel mercury pumps and a good forepump; the gas is carried into two liquid-nitrogen-cooled traps containing activated charcoal.

The starting  $FeC_2O_4$  (0.5-0.8 g.) is placed in the small bulb above the quartz vessel, and the water of crystallization is completely vaporized by heating in vacuum for 12 hours at 200°C. The bulb is turned in the joint, and the  $FeC_2O_4$  drops into the heated lower section of the reactor where it is rapidly decomposed to FeO, CO and CO<sub>2</sub> (the decomposition is complete in about 20 seconds). The product FeO is retained by a quartz wool plug, which must be loose enough to prevent a buildup of pressure during the decomposition.





Fig. 332. Preparation of iron (II) oxide.

stable in the range of 300-560°C and decomposes according to:

$$4 \operatorname{FeO} = \operatorname{Fe}_3 \operatorname{O}_4 + \operatorname{Fe}_3$$

(this decomposition proceeds most rapidly at about 480°C, but ceases below 300°C). The above procedure yields a jet-black

product, readily soluble in dilute acids; it is rapidly oxidized in air, but does not ignite.

II. The preparation from stoichiometric quantities of commercial  $Fe_2O_3$  and reduced iron can also be recommended. The mixture and a few drops of water are sealed into a preevacuated quartz tube, heated for about three days at 900°C, and quenched in cold water.

SYNONYM:

Ferrous oxide.

PROPERTIES:

M.p. 1360°; d 5.7. Crystal structure: B1 (rock salt) type.

REFERENCES:

- I. P. L. Günther and H. Rehaag. Z. anorg. allg. Chem. <u>243</u>, 60 (1939).
- II. R. W. Blue and H. H. Claassen. J. Amer. Chem. Soc. <u>71</u>, 3839 (1949); J. P. Coughlin, E. G. King and K. R. Bonnickson. Ibid. <u>73</u>, 3891 (1951); see also L. Wöhler and R. Günther. Z. Elektrochem. <u>29</u>, 281 (1923).

Iron (II) Hydroxide

### Fe(OH)<sub>2</sub>

The preparation of pure  $Fe(OH)_2$  has been described in detail by Rihl and Fricke as an example of operation under an inert atmosphere. The general experimental arrangement is further described in Part I, p. 72 ff. It consists essentially of a bulb, one side of which can be connected to a high-vacuum pump as well as a source of  $N_2$  or a drying vessel, while the other side is attached to devices for filtration, washing and transfer of products. All operations must be carried out with the most rigorous exclusion of  $O_2$  in an atmosphere of pure, dry  $N_2$ .

The apparatus is first evacuated (high vacuum); then a continuous stream of  $N_2$  is introduced. A centrifuged solution of  $Fe(OH)_2$  (prepared from very pure  $FeCl_2$ ) in conc. aqueous  $NH_3$  is admitted through the filter and diluted with a large quantity of water, causing precipitation of the  $Fe(OH)_2$ . To obtain a denser precipitate (which settles more rapidly), the mixture is heated for about three hours at 80°C and allowed to settle. The mother liquor is filtered off, and the precipitate is washed 10 to 12 times in similar fashion until a positive test for chloride is no longer obtained.

1498

27. IRON

To remove the remaining water, the residue is solidified by immersion of the flask in an ice-salt mixture, full vacuum is applied, and the water is distilled off overnight as the solid slowly melts. The water is condensed in a large trap chilled in Dry Ice-acetone. Complete drying of the product is achieved by keeping it for several additional hours under high vacuum together with a vessel containing  $P_2O_5$ .

PROPERTIES:

Nearly white, slight greenish tinge. When sprayed into air, burns with sparks. Crystal structure: C6 type.

REFERENCE:

S. Rihl and R. Fricke, Z. anorg. allg. Chem. 251, 406 (1943).

## Iron (II, III) Oxide

### Fe<sub>3</sub>O<sub>4</sub>

 $3 \operatorname{Fe_2O_3} + H_2 = 2 \operatorname{Fe_3O_4} + H_2O$ 479.0 22.4 l. 463.0 18.0

Fine  $Fe_2O_3$  powder is heated to 400°C in a large boat placed in a stream of N<sub>2</sub>; then the N<sub>2</sub> is replaced with H<sub>2</sub> saturated with water vapor at 50°C. The connecting tubing between the saturating vessel and the reactor tube must be as short as possible and well insulated against thermal losses. When the reduction is complete, as shown by the disappearance of all of the red  $Fe_2O_{33}$ the product is cooled in a stream of N<sub>2</sub>. Conversion of 10 g. of  $Fe_2O_3$  requires about five hours.

PROPERTIES:

Black, ferromagnetic powder. M.p. 1590°; d 5.11. Mohs hardness 6. Crystal structure: H11 type.

**REFERENCE:** 

S. Hilpert and J. Beyer. Ber. dtsch. chem. Ges. 44, 1608 (1911).

# Iron (III) Hydroxide FeO(OH)

a-FeO(OH)

 $\begin{array}{rrrr} Fe(NO_3)_3 \cdot 9 H_2O &+& 3 NH_3 &=& FeO(OH) &+& 3 NH_4NO_3 &+& 7 H_2O \\ 404.0 & & 51.1 & 88.9 & 240.2 \end{array}$ 

A cold solution of 810 g. of  $Fe(NO_3)_3 \cdot 9H_2O$  in two liters of water is poured slowly, with vigorous stirring, into an amraomia

solution prepared by dissolving 120 g. of gaseous  $NH_3$  in two liters of water (cooling necessary). The hydroxide which precipitates is amorphous to x-ray analysis. It is washed by stirring at least five times with eight-liter portions of cold water, each portion being decanted as completely as possible. The residual slurry is then stirred with sufficient conc. KOH solution to give a mixture approximately 2N and allowed to stand for 3-4 hours. Finally 100°C steam is bubbled through for two hours. The precipitate is thereby transformed completely into bright-yellow  $\alpha$ -FeO(OH), which shows a crystalline x-ray diffraction pattern.

Since removal of the potassium hydroxide by washing is difficult, it is converted to KCl by treatment with somewhat more than the calculated amount of NH<sub>4</sub>Cl. The precipitate is then washed thoroughly with hot water until no further Cl<sup>-</sup> can be detected. The K content is then usually below 0.04%. Drying in a vacuum desiccator affords a product whose water content still exceeds the theoretical (10.14%) by about 2%. Heating in vacuum or in a stream of dry air yields pure  $\alpha$ -Fe<sub>2</sub>O<sub>3</sub>.

The naturally occurring form of  $\alpha$ -FeO(OH), "needle iron ore" or goethite, has an E0<sub>2</sub> type structure.

REFERENCES:

R. Fricke and P. Ackermann. Z. Elektrochem. <u>40</u>, 630 (1934); O. Glemser. Ber. dtsch. chem. Ges. <u>70</u>, 2117(1937); R. Fricke and G. F. Hüttig. Hydroxyde und Oxydhydrate [Hydroxides and Hydrated Oxides], Leipzig, 1937, p. 316; W. Hoppe. Z. Kristallogr. (A) <u>103</u>, 73 (1940).

γ-**FeO(OH**)

$2 \operatorname{FeCl}_2 \cdot 4 \operatorname{H}_2C$	$O + (CH_2)_6 N$	$_{4} + 2 H_{2}O =$	= 2 Fe <b>(</b> OH) <sub>2</sub>	$+ 4 \text{ NH}_{4}\text{Cl}$	$+ 6 H_2O$
397.6	140.1	36.0	179.7	214.0	108.1
Fe(OH) <sub>2</sub>	+ NaNO <sub>2</sub> $+$	HCl = FeO	(OH) + NO	+ NaCl +	$H_2O$
89.9	69.0	36.5 88	3.9 30.0	58.5	18.0

A solution of 120 g. of  $FeCl_2 \cdot 4H_2O$  in three liters of water is filtered, and the filtrate is added to a filtered solution of 168 g. of hexamethylenetetramine (Urotropin) in 600 ml. of water. Bluegreen  $Fe(OH)_2$  is precipitated. Then a solution of 42 g. of NaNO<sub>2</sub> in 600 ml. of water is added with constant stirring; the mixture is heated to about 60°C and allowed to stand three hours (not longer) with occasional agitation. The oxidation, which produces  $\gamma$ -FeO(OH), proceeds with the evolution of considerable quantities of nitrogen oxides. The supernatant liquid is drained off; the precipitate is washed thoroughly with warm water until free of chloride and dried at 60°C in a drying oven. PROPERTIES:

Very friable deep-orange powder comprised of extremely fine needles. Can be converted to pure  $\gamma$ -Fe<sub>2</sub>O<sub>3</sub> by heating in vacuum or in a dry air stream at about 250-400°C. On heating at higher temperatures, or in a sealed tube at 110°C, or even on very intensive grinding, the metastable preparations of the  $\gamma$  series are converted to the stable  $\alpha$  modification.

Occurs in nature as lepidocrocite. Crystal structure:  $E0_4$  type. For metastable  $\beta$ -FeO(OH), see O. Kratky and H. Nowotny, Z. Kristallogr. (A) <u>100</u>, 356 (1938).

REFERENCES:

 O. Glemser. Ber. dtsch. chem. Ges. <u>71</u>, 158 (1938); R. Fricke and W. Zerrweck. Z. Elektrochem. <u>43</u>, 52 (1937); R. Fricke and G. Weitbrecht. Z. anorg. allg. Chem. <u>251</u>, 427 (1943); F. Wagenknecht. Kolloid-Z. <u>112</u>, 35 (1949).

### Iron (III) Oxychloride

### FeOCI

 $FeCl_{s} \cdot 6H_{s}O + 5FeCl_{s} = 6FeOCl + 12HCl$ 270.3 811.1 643.8 37.6

A mixture of 10 g. of  $FeCl_3 \cdot 6H_2O$  and 35 g. of sublimed  $FeCl_3$  is placed in a round, short-neck Pyrex reaction flask, fused on a steam bath, and allowed to solidify; then 15 additional g. of  $FeCl_3$  is added. The reaction indicated above takes place when the mass is heated to 250-300°C (maximum). It is best to immerse the open flask rather deeply in an oil bath held at 250°C, so that no moisture condenses at the neck. The reaction is complete after 60-80 minutes, when no further evolution of HCl is observed.

The reaction mass, which converts to a solid red cake, is cooled and pulverized, washed briefly with a large amount of cold water and then with acetone (to remove excess  $FeCl_3$ ), and dried in vacuum.

PROPERTIES:

Rust-colored powder consisting of small red needles. Free of  $Fe_2O_3$  if made from sublimed  $FeCl_3$  and if reaction temperatures no higher than 300°C are used. Disproportionates above 300°C into the oxide and chloride. Crystal structure:  $EO_5$  type.

REFERENCE:

H, Schäfer. Z. anorg. Chem. 260, 279 (1949).
# Iron (II) Sulfide FeS

Fe + S = FeS55.8 32.1 87.9

Pure FeS of stoichiometric composition is obtained from pure reduced Fe and distilled S. Exactly weighed quantities of the two substances are sealed in a quartz tube evacuated in high vacuum, and are heated for about 24 hours at 1000°C; at higher temperatures, the quartz tube bursts. One then tests for completeness of the conversion: the reaction is complete if S no longer collects at that end of the hot tube which is cooled for test purposes. The lustrous gray product obtained is somewhat sintered and readily pulverized.

SYNONYM:

Ferrous sulfide.

PROPERTIES:

M.p. 1195°; d 4.84. Crystal structure: B8 type.

**REFERENCES:** 

H. Haraldsen, Z. anorg. allg. Chem. <u>231</u>, 81 (1937); G. Hägg and J. Sucksdorff. Z. phys. Chem. (B) <u>22</u>, 444 (1933).

### Iron Nitrides

### Fe<sub>2</sub>N, Fe<sub>4</sub>N

I.

Fine  $Fe_2O_3$  powder is placed in a porcelain boat at 500°C and reduced as completely as possible with H<sub>2</sub>; then, without allowing any air to penetrate, NH<sub>3</sub> is introduced at 350-550°C until the H<sub>2</sub> content of the exit gas decreases to a low, constant level. The product is then allowed to cool in the stream of NH<sub>3</sub>. It corresponds in composition to the formula Fe<sub>2</sub>N (theoretically 11.1% N), has the structure of Hägg's  $\zeta$  phase, and exhibits a very narrow region of homogeneity. Often, however, the same composition yields a mixed crystal (solid solution) phase  $\epsilon$ . On heating in vacuum at about 500°C the product is converted (with loss of N<sub>2</sub>) first to the solid solution  $\epsilon$  and then to Fe<sub>4</sub>N ( $\gamma'$  phase, theoretically 5.9% N). II. Another method for preparing  $Fe_4N$  consists in heating reduced iron in the presence of an appropriate mixture of  $H_2$  and  $NH_3$ .

PROPERTIES:

Fe<sub>2</sub>N: Formula weight 125.7; d 5.02.

Fe<sub>4</sub>N: Formula weight 237.4; d 6.57 (?). Crystal structure: L 10 type.

REFERENCES:

- I. G. Hägg. Z. phys. Chem. (B) 8, 455 (1930); also E. Lehrer. Z. Elektrochem. <u>36</u>, 388, 460 (1930); O. Eisenhut and E. Kaupp. Ibid. <u>36</u>, 394 (1930); S. Satoh. Bull. Chem. Soc. Japan <u>7</u>, 315 (1932); abstract in Chem. Zentr. <u>1933</u>, I, 752.
- II. St. Brunauer, M. E. Jefferson, P. H. Emmett and S. B. Hendricks. J. Amer. Chem. Soc. <u>53</u>, 1778 (1931); Ch. Guillaud and H. Creveaux. Comptes Rendus Hebd. Séances Acad. Sci. <u>222</u>, 1170 (1946); H. W. Köhlschütter and M. Pavel. Z. anorg. allg. Chem. <u>255</u>, 65, 73 (1947).

# Iron Carbide

# Fe<sub>3</sub>C

It is best to start with electrolytic iron sheet; this is held over benzene vapor to deposit a layer of carbon, then baked for a long time in vacuum at 700°C and slowly cooled. To isolate  $Fe_3C$  the sheet is placed in a neutral  $FeCl_2$  bath and used as an anode at the lowest possible current density. The bath is the same as used in the preparation of electrolytic iron. In this way very pure  $Fe_3C$  is left behind as a coarsely crystalline gray powder. It is washed with dilute acetic acid, water, alcohol and ether, and dried in vacuum.

Alternate methods: a) Iron carbide may also be isolated quantitatively while measuring the anode potential; see E. Houdremont, P. Klinger and G. Blaschczyk, Techn. Mitt. Krupp, Forschungsber. <u>4</u>, 311; Arch. Eisenhüttenw. <u>15</u>, 257 (1941).

b) Solution of white, low-Si pig iron in 1 N acetic acid; see O. Ruff and E. Gersten, Ber. dtsch. chem. Ges. 45, 64 (1912).

c) For the preparation of almost pure fused  $\overline{Fe}_{3}C$ , see F. Wever, Mitt. KWI Eisenforschung <u>4</u>, 67 (1923).

SYNONYM:

Cementite.

PROPERTIES:

Formula weight 179.52. Hardness 3.2-3.3; d (15°) 7.66. Crystal structure:  $D0_{11}$  type.

The decomposition of Fe<sub>3</sub>C to Fe and C proceeds at a high rate only above  $1050^{\circ}$ C. Below  $500^{\circ}$ C a carbide having the formula Fe<sub>2</sub>C may also be obtained in various modifications [L. J. E. Hofer, E. M. Cohn and W. C. Peebles, J. Amer. Chem. Soc. <u>71</u>, 189 (1949); H. H. Podgurski, J. T. Kummer, T. W. de Witt and P. H. Emmett, ibid. 72, 5382 (1950)].

**REFERENCE:** 

G. Naeser. Mitt. KWI für Eisenforschung 16, 211 (1934).

### Lithium Ferrate (III)

### LiFeO<sub>2</sub>

Obtained by heating to high temperature or fusion of an intimate mixture of  $\text{Li}_2\text{CO}_3$  with fine  $\text{Fe}_2\text{O}_3$  powder derived from  $\text{FeC}_2\text{O}_4$ . At temperatures above about 670°C, the stable modification is of the B1 type, with random distribution of the metal atoms; annealing at 570°C converts it to a nearly cubic tetragonal modification with ordered distribution.

Alternate method: Heating conc. LiOH solution with  $Fe_2O_3$  at 600°C under pressure [E. Posniak and T. F. W. Barth, Phys. Rev. <u>38</u>, 2234 (1931)].

**REFERENCE:** 

F. Barblan, E. Brandenberger and P. Niggli. Helv. Chim. Acta <u>27</u>, 88 (1944).

### Potassium Ferrate (VI)

K<sub>2</sub>FeO<sub>4</sub>

I. BY OXIDATION WITH NaOC1:

 $2 \operatorname{NaOH} + \operatorname{Cl}_2 = \operatorname{NaCl} + \operatorname{NaOCl} + \operatorname{H}_2 O$ 70.9 58.5 74.5  $3 \operatorname{NaOCI} + 2 \operatorname{Fe}(\operatorname{NO}_3)_3 \cdot 9 \operatorname{H}_2O + 10 \operatorname{NaOH} =$ 223.5808.0 400.0  $2 \operatorname{Na}_{2}\operatorname{FeO}_{4} + 6 \operatorname{NaNO}_{3} + 3 \operatorname{NaCl} + 23 \operatorname{H}_{2}O$ 331.7 510.0 175.5  $Na_2FeO_4 + 2 KOH = K_2FeO_4 + 2 NaOH$ 165.9112.2 198.0 80.0

A solution of 30 g. of NaOH in 75 ml. of water is cooled, and  $Cl_2$  is admitted with vigorous stirring until a weight increase of

1504

20 g. is recorded. The cooling rate should be such that the mixture temperature does not rise above  $20^{\circ}$ C. Then 70 g. of solid NaOH is added slowly and with constant stirring; the temperature is allowed to rise to  $25-30^{\circ}$ C to speed the dissolution. As soon as this is complete, the solution is cooled once again to  $20^{\circ}$ C and passed through a glass filter to remove the precipitated NaCl.

With the NaOCl solution held at 25-30 °C, 25 g. of solid Fe(NO<sub>3</sub>)<sub>3</sub>  $\cdot$  9 H<sub>2</sub>O is slowly added (stirring), the mixture is saturated with solid NaOH at 30 °C, and the ferrate (VI) solution is either filtered through a coarse glass filter or, better, centrifuged.

The precipitation of  $K_2FeO_4$  is carried out in a 250-ml. beaker by addition of 100 ml. of saturated KOH while stirring and cooling to 20°C. After precipitation, stirring is continued for five minutes; the compound is collected on a medium-porosity glass filter and redissolved by treatment with four or five 10-ml. portions of 3 M KOH. The solutions are combined in a 250-ml. beaker and 50 additional ml. of saturated KOH is added; after five minutes of stirring at 20°C the solution, which is now about 11 M in KOH, is filtered through a medium-porosity glass filter.

For further purification, the  $K_2FeO_4$  collected on the filter is treated with 10 ml. of benzene, then three to five times with 20-ml. portions of 95% ethanol (aldehyde-free), and finally stirred for 20 min. (in a large beaker) with 1000 ml. of 95% ethanol. This last treatment is repeated three additional times. The product is collected on a glass filter, washed in the absence of atmospheric moisture with 50 ml. of ether, and dried in a vacuum desiccator.

The yield is 45-75% of theoretical, the purity 92-96%. One further reprecipitation from 6 M KOH raises the purity to 98.5-99%.

II. ELECTROLYTIC METHOD :

The electrolysis is conducted in a cylindrical vessel (95 mm. I. D. and 100 mm. high) in which a porous clay cell (50 mm. I. D. and 80 mm. high) is set. The cell is held in place by a paraffinsoaked cork ring in such a way that it touches the bottom of the cylinder. The anode is a strip of transformer iron sheet (0.3%Mn;  $27 \times 3.7$  cm. = 100 cm.<sup>2</sup> of surface on each side) which adheres closely to the outer wall. The cathode consists of a screen of iron wire rolled into a cylinder and placed inside the cell. The anode is welded to a thick, acetylene-flame-cleaned iron wire, which passes through the cork ring and is sealed in with picels wax. The cork ring also carries a thermometer, a two-bulb pressure release tube containing 30% KOH, and a short tube (7 mm. I. D.) for removing samples.

Both cathode and anode are etched with 1:1 HCl shortly before the start of the experiment, and are then rinsed with water. The cylinder is now filled with 200 ml. and the cell with 60 ml. of freshly prepared 40% NaOH precooled to 25-28 °C. The apparatus is assembled and cooled externally with ice water. To start with, the electrode intended as the anode is connected as the cathode and the electrolysis conducted for 3-5 min. at 3.5 amp. and 110 v. d. c. The resistance in the circuit should be about 30 ohms. The polarity is then reversed; the actual electrolysis takes four hours at 4.5 amp. (approximately 5.8 v.); the temperature in the anode space must never exceed 35 °C.

The current is shut off and the anode electrolyte freed of traces of  $Fe(OH)_3$  by centrifuging or filtering rapidly through a mediumporosity glass filter. The filtrate is cooled to 10°C, and 75 g. of KOH pellets is added with continued cooling and vigorous shaking. An additional hour at 0°C is allowed for completion of the reaction. The K<sub>2</sub>FeO<sub>4</sub> precipitate is rapidly collected on a medium-porosity glass funnel, washed at once with ice-cold absolute methanol, and dried in vacuum over P<sub>2</sub>O<sub>5</sub>. The anode electrolyte, which after four hours is about 0.15 M in ferrate (VI), affords about 5 g. of K<sub>2</sub>FeO<sub>4</sub> with a purity of 95%. The main impurity is carbonate; in addition there is 0.1% Mn, as well as compounds passing into the solution from the earthenware cell. These could probably be avoided by using a cell made of polytetrafluoroethylene or a similar material. The current efficiency is about 25%.

#### PROPERTIES:

Small lustrous crystals, very dark violet to black; stable only when completely dry. Readily soluble in water; concentrated solutions decompose rapidly; very dilute solutions are much more stable. Chloride ions and FeO(OH) markedly accelerate the decomposition. The instability of solutions as a function of pH is described by J. M. Schreyer and L. T. Ockerman, Anal. Chem. 23, 1312 (1951).

BEFE HENCE.

 G. W. Thompson, L. T. Ockerman and J. M. Schreyer. J. Amer. Chem. Soc. <u>73</u>, 1379 (1951); H. J. Hrostowski and A. B. Scott. J. Chem. Phys. <u>18</u>, 105 (1950); B. Helferich and K. Lang. Z. anorg. Chem. <u>263</u>, 171 (1950); R. Scholder, H. von Bunsen, F. Kindervater and W. Zeiss. Z. anorg. allg. Chem. <u>282</u>, 268 (1955); L. Moeser. J. prakt. Chem. [2] <u>56</u>, 431 (1897). II. G. Grube and H. Gmelin. Z. Elektrochem. <u>26</u>, 160 (1920); modified directions based on unpublished experiments made together with H. Noeth.

## Potassium Iron (III) Sulfide

#### KFeS<sub>2</sub>

An intimate mixture of 5 g. of Fe powder (obtained by reduction with  $H_2$ ), 25 g. of  $K_2CO_3$ , 5 g. of  $Na_2CO_3$ , and 30 g. of S is slowly heated in a half-filled covered porcelain crucible until the mass flows smoothly; it is then held at bright-red heat for about one hour. A better method consists in heating the mixture (which is placed in a boat inserted in a porcelain tube) under nitrogen at 900-1000°C. In either case the crucible is cooled slowly and broken. The fragments of the melt are soaked in warm water, the solution being frequently decanted, until the only substance remaining in the flask is the reddish violet needles of KFeS<sub>2</sub> (semimetallic luster). If a considerable quantity of colloidal, amorphous product is obtained, the reaction temperature was not sufficiently high. The crystals are washed with water and alcohol and dried as rapidly as possible at 100°C. The yield is 12-14 g.; the theoretical yield, based on Fe, is 14.25 g.

PROPERTIES:

Formula weight 159.1. Insoluble in water; indefinitely stable in dry air.

REFERENCES:

K. Preis. J. prakt, Chem. <u>107</u>, 12 (1869); R. Schneider. Ibid. <u>108</u>, 16 (1869).

**Basic Iron (III) Sulfate** 

 $\operatorname{Fe}_{3}(\operatorname{SO}_{4})_{2}(\operatorname{OH})_{5} \cdot 2 \operatorname{H}_{2}O \text{ or } 3 \operatorname{Fe}_{2}O_{3} \cdot 4 \operatorname{SO}_{3} \cdot 9 \operatorname{H}_{2}O$ 

 $3 \operatorname{Fe}_{2}(\operatorname{SO}_{4})_{3} + 14 \operatorname{H}_{2}O = 2 \operatorname{Fe}_{3}(\operatorname{SO}_{4})_{2}(\operatorname{OH})_{5} \cdot 2 \operatorname{H}_{2}O + 5 \operatorname{H}_{2}\operatorname{SO}_{4}$   $\frac{1}{1_{10}}: 120.0 \quad 25.2 \quad 96.1 \quad 49.0$ 

The basic sulfate, which is stable over a large temperature range (up to  $170^{\circ}$ C), is readily obtained in crystalline form by

heating a sealed tube containing an approximately 20% solution of  $Fe_2(SO_4)_3$  at 150°C. The product is a fine, orange-yellow powder consisting of small, transparent, cubelike rhombohedra.

REFERENCES:

 E. Posniak and H. E. Merwin, J. Amer. Chem. Soc. <u>44</u>, 1965 (1922);
 N. V. Shishkin, Zh. Obshch, Khim. <u>21</u>, 456(1951); Athanasesco, Comptes Rendus Hebd. Seances Acad. Sci. <u>103</u>, 271 (1886).

#### **Basic Iron (III) Acetate**

#### $[Fe_3(CH_3COO)_6(OH)_2]CH_3COO \cdot H_2O$

 $3 \text{ Fe}(\text{OH})_{2} + 7 \text{ CH}_{3}\text{COOH} = [\text{Fe}_{3}(\text{CH}_{3}\text{COO})_{6}(\text{OH})_{2}]\text{CH}_{3}\text{COO} \cdot \text{H}_{2}\text{O} + 6 \text{ H}_{2}\text{O}$ 320.6 420.2 632.7

The monoacetate of the complex base, which occurs predominantly in the monoacidic form, crystallizes readily from dilute acetic acid (about 10% or 1.6 N). The triacetate is formed only from solutions containing at least 65% acetic acid by weight (11.4 N); these are allowed to stand in a vacuum desiccator over conc.  $H_2SO_4$ .

To prepare the monoacetate, a very dilute  $FeCl_3$  solution is treated at room temperature with aqueous  $NH_3$ ; the precipitate is washed for several days with cold water, which is frequently decanted, and washed thoroughly again on a filter. The slurry of hydrated iron oxide thus obtained is dissolved (heating) in about an equal amount of acetic acid. Crystallization takes place if the solution is allowed to stand in an open dish for several days in a well-ventilated spot. After filtration, the salt is kept for some time over soda lime to absorb the acetic acid.

SYNONYM:

Triiron (III) hexaacetatodihydroxomonoacetate.

PROPERTIES:

Transparent, brick-red rhombic leaflets. Dissolves slowly in cold water, rapidly in hot water. Only sparingly soluble in acetic acid.

**REFERENCES:** 

 A. Krause. Z. anorg. allg. Chem. <u>169</u>, 286 (1928); R. Weinland and E. Gussmann. Ber. dtsch. chem. Ges. <u>42</u>, 3888(1909); Z. anorg. Chem. <u>66</u>, 157 (1910).

1508

## Hexacyanoferric (II) Acid

### H<sub>4</sub>Fe(CN)<sub>6</sub>

 $\begin{array}{rrrr} K_{4}Fe(CN)_{6} \cdot 3 \ H_{2}O &+ \ 4 \ HCl \ + \ 2 \ (C_{2}H_{5})_{2}O \ = \\ & 422.4 & 145.9 & 148.2 \\ H_{4}Fe(CN)_{6} \cdot 2 \ (C_{2}H_{5})_{2}O \ + \ 4 \ KCl \ + \ 3 \ H_{2}O \\ & 364.1 & 298.2 \end{array}$ 

A solution of 42 g. of  $K_4Fe(CN)_6 \cdot 3H_2O$  in 350 ml. of water is mixed with 100 ml. of conc. HCl (d 1.19); any KCl which separates is redissolved by addition of some water. After thorough chilling, about 50 ml. of ether is added. The etherate separates in several hours as colorless tablets. These are filtered off, washed with a small quantity of dilute HCl containing some ether, and redissolved in 50 g. of alcohol. After residual undissolved KCl is removed by filtration, the compound is reprecipitated by addition of 50 g. of ether, filtered off, and washed with ether. It is finally transferred to a round flask and converted to  $H_4Fe(CN)_6$  by heating at 40-50°C in aspirator vacuum.

SYNONYMS:

Hydrogen hexacyanoferrate (II), ferrocyanic acid.

PROPERTIES:

Snow-white when pure.  $d(25^{\circ})$  1.536. Indefinitely stable if dry; gradually becomes blue in moist air. Elimination of HCN begins at about 100°. Readily soluble in water or alcohol; insoluble in ether or acetone. The bright lemon-yellow aqueous solution decomposes on heating or in light. Solubility (14°) 13 g./100 g. of aqueous solution.

**REFERENCES:** 

W. Biltz. Z. anorg. allg. Chem. <u>170</u>, 161 (1928); A. Mittasch and E. Kuss. Z. Elektrochem. <u>34</u>, 159 (1928).

## Ammonium Hexacyanoferrate (II)

## (NH<sub>4</sub>)<sub>4</sub>Fe(CN)<sub>6</sub>

 $H_4Fe(CN)_6 + 4 NH_3 = (NH_4)_4Fe(CN)_6$ 216.0 68.1 284.1

To prepare a completely potassium-free salt, a conc. aqueous solution of pure hexacyanoferric (II) acid is neutralized with pure

10% aqueous NH<sub>3</sub>; the salt which precipitates is collected, washed several times with alcohol and then with ether.

PROPERTIES:

Readily soluble in water, insoluble in alcohol. In vacuum, decomposition begins above 100°C.

REFERENCE:

A. Mittasch and E. Kuss. Z. Elektrochem. <u>34</u>, 59 (1928).

### Hexacyanoferric (III) Acid

#### $H_3Fe(CN)_6$

# $K_3Fe(CN)_6$ + 3 HCl = H<sub>3</sub>Fe(CN)\_6 + 3 KCl 329.3 109.4 214.9 223.7

Cold saturated  $K_3Fe(CN)_6$  solution (40 ml.) is treated slowly and in the cold with 40 ml. of fuming HCl; the mixture is allowed to stand in an ice bath with frequent agitation for about half an hour. The KCl precipitate is removed by filtration and the filtrate is shaken with 70 ml. of ether. Three layers are formed: aqueous, oily and ethereal. After draining the aqueous layer, the middle, oily layer is allowed to clarify. It is then separated from the ether layer and the oil is completely freed of ether under vacuum. This results in crystallization of a yellow etherate of  $H_3Fe(CN)_6$ ; finally, however, pure  $H_3Fe(CN)_6$  remains as a brown mass. The acid may be recrystallized by solution in absolute ethanol and evaporation of the solvent. The compound must not be allowed to contact metal or rubber and should be kept as dry as possible.

SYNONYM:

Hydrogen hexacyanoferrate (III), ferricyanic acid.

PROPERTIES:

Rather readily soluble in water and alcohol; unlike  $H_4Fe(CN)_6$ , it is also soluble in ether-alcohol mixtures. The aqueous solution is yellow to brown.

REFERENCE:

W. M. Cumming and D. G. Brown. J. Soc. Chem. Ind. Trans. <u>44</u>, 110 (1925).

1510

### Sodium Hexathiocyanoferrate (III)

Na<sub>3</sub>Fe(SCN)<sub>6</sub> · 12 H<sub>2</sub>O

 $Fe(OH)_{3} + 3 HSCN + 3 NaSCN + 9 H_{2}O = Na_{3}Fe(SCN)_{6} \cdot 12 H_{2}O$ 106.9 177.2 243.2 689.4

Aqueous HSCN solution is added in the cold to a known amount of freshly precipitated and well washed hydrated iron (III) oxide until the solid is barely dissolved. Then NaSCN is added until about nine moles of NaSCN are present for each mole of Fe(SCN)<sub>3</sub>. The solution is allowed to stand in a desiccator over conc.  $H_2SO_4$ for several weeks; very dark-red crystals, which exhibit an intense green color when viewed by reflected light, separate out. On further evaporation of the mother liquor, precipitation of NaSCN also begins; on heating, the Fe<sup>3 +</sup> is partly reduced.

SYNONYM:

Sodium iron (III) thiocyanate.

PROPERTIES:

Deliquescent in air; converted to the trihydrate by storage over conc.  $H_2SO_4$ . Soluble in alcohol, giving a permanganatelike violet color; may be recrystallized from such solutions without decomposition.

**REFERENCES:** 

G. Krüss and H. Moraht. Liebigs Ann. <u>260</u>, 209 (1890); A. Rosenheim and R. Cohn. Z. anorg. Chem. <u>27</u>, 295 (1901); H. I. Schlesinger and H. B. van Valkenburgh. J. Amer. Chem. Soc. <u>53</u>, 1215 (1931).

## Sodium Pentacyanoamminoferrate (II)

## $Na_3[Fe(CN)_5NH_3] \cdot 3H_2O$

$$Na_{2}[Fe(CN)_{5}NO] \cdot 2 H_{2}O + 2 NH_{3} + NaOH$$
298.0 34.1 40.0

$$= Na_{3}[Fe(CN)_{5}NH_{3}] \cdot 3H_{2}O + N_{2} + H_{2}O$$
326.0

A mixture of 30 g. of  $Na_2[Fe(CN)_5NO] \cdot 2H_2O$  with 120 ml. of water is prepared and cooled in an ice-salt mixture. Ammonia is then introduced at +10°C until saturation; the temperature during

this operation must not exceed 20°C. The solution is allowed to remain under a loose cover for several hours (not longer) at 0°C, and the crystals which separate from the deep brownishyellow solution are collected by filtration. The remainder of the compound can be precipitated from the solution by addition of CH<sub>3</sub>OH. The product may be purified by solution in some cold water, from which it is precipitated as fine, bright-yellow needles by careful addition of 90% alcohol.

The hexahydrate is hygroscopic and readily loses  $NH_3$ . It is dried to constant weight by storing for several days in a vacuum desiccator over  $CaCl_2$ . It then contains three (or, according to Hölzl, 2.5) moles of  $H_2O$ . Yield: 24 g. (73%). The yellow aqueous solution decomposes on heating, precipitating the hydroxide.

REFERENCES:

K. A. Hofmann. Z. anorg. Chem. <u>10</u>, 264 (1895); W. Manchot, E. Merry and P. Woringer. Ber. dtsch. chem. Ges <u>45</u>, 2876 (1912);
 F. Hölzl and K. Rokitansky. Monatsh. Chem. <u>56</u>, 82 (1930).

# Sodium Pentacyanoamminoferrate (III)

# Na<sub>2</sub>[Fe(CN)<sub>5</sub>NH<sub>3</sub>] · H<sub>2</sub>O

$Na_{3}[Fe(CN)_{5}NH_{3}] \cdot 3 H_{2}O + NaNO_{2} + CH_{3}COOH =$								
326.0		69.0		6	0.0			
$Na_{2}[Fe(CN)_{5}NH_{3}] \cdot H_{2}O$	+	NO	+	NaOH	+	CH <sub>3</sub> COONa	+	$2 H_2O$
266.9		30.0		40.0		82.0		

A solution of 20 g. of NaNO<sub>2</sub> in 50 ml. of water is treated at  $0^{\circ}$ C with 20 ml. of 30% acetic acid and then with 30 g. of Na<sub>3</sub>[Fe-(CN)<sub>5</sub>NH<sub>3</sub>] · 3 H<sub>2</sub>O. After two hours, 1:1 alcohol-ether is added; this first precipitates a violet aquo complex salt (formed in a side reaction); further addition of the alcohol-ether mixture precipitates the desired salt. This is purified by repeated solution in cold water and reprecipitation with alcohol. The dark-yellow powder is dried to constant weight in vacuum over conc. H<sub>2</sub>SO<sub>4</sub>; it then contains one or two moles of water.

The salt dissolves readily inwater, giving a brownish-red color.

**REFERENCES:** 

K. A. Hofmann. Liebigs Ann. <u>312</u>, 24 (1900); F. Hölzl and K. Rokitansky. Monatsh. Chem. <u>56</u>, 82 (1930).

# SECTION 28

Cobalt, Nickel

O. GLEMSER

## Metallic Cobalt

I. Prepared by reduction of precipitated cobalt oxalate with hydrogen.

Cobalt oxalate, precipitated in the cold, is dried at  $120^{\circ}$ C and ground to a fine powder. It is then reduced with H<sub>2</sub> (six hours at 500°C), with the temperature being raised rapidly at the beginning of the run. The product is cooled, ground and reduced once more; the fine metal powder is stored in a glass vessel under alcohol.

### II. VERY PURE COBALT

Impurities, principally Fe, Cu and Ni, are removed by various precipitation reactions and by electrolysis. Finally, very pure Co is deposited electrolytically from a  $CoSO_4$  solution. According to Kershner, Hoertel and Stahl, the pure metal still contains 0.001-0.002% Ni, 0.001-0.003% Fe, a maximum of 0.001% Cu, and 0.005% S.

### A) REMOVAL OF IMPURITIES

A solution of the cobalt (II) salt is treated with  $Na_2CO_3 \cdot 10 H_2O$ , added in small portions and with stirring, until the pH reaches 3.5; then 1 N  $Na_2CO_3$  solution is added until the cobalt carbonate just barely precipitates. The precipitated carbonates are removed by filtration. Following this, 8 ml. of a saturated aqueous solution of 1,2-cyclohexanedione dioxime is added for each 10 mg. of Ni, Cu and Fe present in the filtrate. The Ni precipitates out; the suspension is heated for one hour at 90-95°C with occasional stirring, and the precipitate is filtered off.

The solution, which should now contain about 10% Co, is adjusted to a pH of 5.5 with  $H_2SO_4$  or  $Na_2CO_3$ . Then, 0.2 moles of pyridine is added per liter of solution; the latter is electrolyzed

at  $25^{\circ}$ C (stirring), using a mercury cathode ( $206 \text{ cm.}^2$  of surface area) and a platinum sheet anode. The cathode potential stays constant at 0.78 volt. At a Ni content of 0.05-0.10 g./liter of solution, the electrolysis requires eight hours; at a level of 0.5 g. of Ni/liter, it takes 24-32 hours.

After the electrolysis, the solution is filtered and the  $Co(OH)_2$ precipitated from the filtrate by addition of NaOH. The precipitate is washed by decantation with hot distilled water until the odor of pyridine is no longer apparent. The precipitate is then filtered off and dissolved in dilute  $H_2SO_4$  in such a way as to give a solution containing 90-100 g. of Co. This solution then constitutes the starting material for the electrolytic separation of the metal (see the following).

B) SEPARATION OF THE METAL

The electrolysis cell is a 4-liter beaker with a side arm through which the solution can overflow into a collecting vessel. To avoid contamination from external sources, the cell and collecting vessel are placed in a glass cabinet. The solution is passed through the cell at 1-2 liters per hour, the pH being maintained at 1.2-1.6 by addition of the pyridine-free  $CoSO_4$  solution. The electrolysis conditions are: 50-55°C, three platinum anodes and two titanium cathodes (the latter having about the same surface area as the anodes), cathode current density 40 amp./in.<sup>2</sup> The deposit of highly purified cobalt can be removed from the titanium cathodes.

To avoid accumulation of impurities in the electrolyte, the latter is periodically (as the need arises) reprocessed according to the procedure given in subsection A on removal of impurities (filtration, precipitation, washing and redissolution in  $H_2SO_4$ ).

Alternate method: Reduction of CoO or  $Co_3O_4$  with H<sub>2</sub> for five hours at 500°C (Gmelin, 8th ed., volume on cobalt).

### PROPERTIES:

Atomic weight 58.94. Black metallic powder. M.p. 1492°C, b.p. 3183°C. Ferromagnetic. Readily soluble in dil.  $HNO_3$ ; passivated by conc.  $HNO_3$ .

### REFERENCES:

- I. W. Biltz. Z. anorg. allg. Chem. <u>134</u>, 25 (1924); G. F. Huttig and R. Kassler. Z. anorg. allg. Chem. <u>187</u>, 24 (1930).
- II. K. K. Kershner, F. W. Hoertel and J. C. Stahl. U. S. Dept. Interior, Bur. Mines Rep. Invest. <u>1956</u>, I (Chem. Zentr. <u>1957</u>, 255).

# Cobalt (II) Chloride

## CoCl<sub>2</sub>

I.  $CoCl_2 + 6 SOCl_2 = CoCl_2 + 12 HCl + 6 SO_2$ (6H<sub>2</sub>O) 238.0 713.9 129.9 437.6 384.4

Fine  $CoCl_2 \cdot 6H_2O$  powder is placed in a flask provided with a ground joint and is covered with  $SOCl_2$ . The mixture is refluxed for several hours. The excess  $SOCl_2$  is then evaporated on a steam bath. The  $SOCl_2$  which clings to the product is removed by repeated evacuation of the flask.

II. 
$$C_0(CH_3COO)_2 \cdot 4H_2O + 6CH_3COCl =$$
  
249.1 471.0  
 $C_0Cl_2 + 2(CH_3CO)_2O + 4CH_3COOH + 4HCl$   
129.9 204.2 240.2 145.9

A Pyrex tube  $18 \times 200$  mm. is charged with 4.0 g. of fine  $Co(CH_3COO)_2 \cdot 4H_2O$  powder. It is then closed off with a rubber stopper; the latter carries a dropping funnel and a fritted-glass filtering finger. Then, 15 ml. of benzene is added with agitation (magnetic stirrer), followed by  $CH_3COCl$  (slow addition until about 10% excess). The mixture is stirred for 30 minutes, the  $CoCl_2$  precipitate allowed to settle, and the mother liquor siphoned off through the filtering finger. The residue is treated with benzene and  $CH_3COCl$  to complete the reaction. The supernatant liquid is removed by filtration and the  $CoCl_2$  is washed three or four times with anhydrous benzene; it is then dried for two hours at 150°C under nitrogen.

Alternate methods: a) Heating of  $CoCl_2 \cdot 6H_2O$  with  $COCl_2$ in a sealed tube at 200°C [H. Hecht (1947)].

b) Heating of  $CoCl_2 \cdot 6 H_2O$  in a stream of dry HCl at 160-170 °C [G. L. Clark, A. J. Quick and W. D. Harkins, J. Amer. Chem. Soc. 42, 2483 (1920)]. Simple heating at 140 °C yields a somewhat basic salt.

PROPERTIES:

Leaflets, colorless in very thin layers, pale blue in layers over 1 mm. thick. M.p. 735°C, b.p. 1049°C;  $d_4^{25}$  3.367. Heat of formation (25°C): -77.8 kcal./mole. Decomposes on long heating in air at 400°C. Sublimes at 500°C in HCl gas, forming loose crystalline fragments. Hygroscopic. Solubility in H<sub>2</sub>O (g. of CoCl<sub>2</sub>/100 g. of solution): 29.5 (0°C); 34.86 (20°C); 51.93 (98°C). Soluble in methanol, ethanol, acetone, pyridine and ether. Crystal structure: C19 type.

**REFERENCES:** 

- I. H. Hecht. Z. anorg. Chem. 254, 51 (1947).
- II. G. W. Watt, P. S. Gentile and E. P. Helvenston. J. Amer. Chem. Soc. <u>77</u>, 2752 (1955).

# Hexaamminecobalt (II) Chloride

# $[Co(NH_3)_6]Cl_2$

$CoCl_2$	+	$6 \ NH_3$	=	$[Co(NH_3)_6]Cl_2$
(6H <sub>2</sub> O) 238.0		102.2		232.0

A mixture of 15 g. of  $CoCl_2 \cdot 6 H_2O$  and 14 ml. of  $H_2O$  is heated to boiling in the absence of air and treated hot with sufficient conc. NH<sub>3</sub> to produce complete solution; the solution is then filtered. Deaerated alcohol (air boiled out under reflux) is added to the hot filtrate until a permanent clouding is just barely obtained. The solution is cooled in running water and the solid thus precipitated is filtered off. It is washed with 1:1 conc. NH<sub>3</sub>: alcohol, then with the same mixture in 1:2 ratio, and lastly with deaerated, NH<sub>3</sub>saturated alcohol. The product is dried over KOH in a high vacuum. Yield: 7g.

The precipitation, washing, filtration and drying must be carried out in an  $O_2$ -free atmosphere (for technique see Part I).

Alternate method: Passage of  $NH_3$  over  $CoCl_2$  at room temperature [W. Biltz and B. Fetkenheuer (1914)].

PROPERTIES:

Flesh-colored powder or rose-red crystals. In high vacuum over  $H_2SO_4$  (65-67°C), converts to blue *trans*-[CoCl<sub>2</sub>(NH<sub>4</sub>)<sub>2</sub>]. Relatively stable to  $O_2$  when dry; gradually oxidized in air when moist. Readily soluble in dil. ammonia, sparingly soluble in conc. ammonia, insoluble in alcohol.  $d_4^{25}$  1.479. Crystal structure: J1<sub>1</sub> (K<sub>2</sub>PtCl<sub>6</sub>) type.

**REFERENCE:** 

W. Biltz and B. Fetkenheuer. Z. anorg. allg. Chem. 89, 97 (1914).

1516

# Cobalt (II) Bromide

### CoBr<sub>2</sub>, CoBr<sub>2</sub> · 6 H<sub>2</sub>O

#### ANHYDROUS CoBr<sub>2</sub>

I.

 $CoBr_2 \cdot 6 H_2O = CoBr_2 + 6 H_2O$ 326.9 218.8 108.1

Prepared by careful heating of  $CoBr_2 \cdot 6H_2O$  to 130-150°C, followed by sublimation in high vacuum at 500°C.

II. Treatment of  $Co(CH_3COO)_2 \cdot 4H_2O$  with  $CH_3COBr$  in benzene in a manner analogous to the preparation of  $CoCl_2$ .

Alternate methods: a) Heating of CoBr<sub>2</sub> · 6H<sub>2</sub>O in a stream of HBr at 500°C [G. Crut. Bull. Soc. Chim. France [4] <u>35</u>, 550 (1924)].

b) Allowing CoBr<sub>2</sub> · 6H<sub>2</sub>O to stand for one week over conc. H<sub>2</sub>SO<sub>4</sub> [G. L. Clark and H. K. Bruckner (1922)].

#### PROPERTIES:

Green solid or lustrous green crystalline leaflets. M.p.  $678^{\circ}$  (under HBr and N<sub>2</sub>); $d_2^{25}$ 4.909. Heat of formation: -63.8 kcal./mole. Hygroscopic; in air, transforms to CoBr<sub>2</sub> · 6 H<sub>2</sub>O. Readily soluble in H<sub>2</sub>O (red color). Saturated aqueous solution contains 66.7 g. of CoBr<sub>2</sub> at 59°C, 68.1 g. at 97°C (per 100 g. solution). Readily soluble in methanol, ethanol, acetone and methyl acetate; sparingly soluble in tetranitromethane. Crystal structure: C 6 type.

**REFERENCES:** 

G. L. Clark and H. K. Bruckner. J. Amer. Chem. Soc. <u>44</u>, 230 (1922); W. Biltz and E. Birk. Z. anorg. allg. Chem. <u>127</u>, 34 (1923); G. W. Watt, P. S. Gentile and E. P. Helveston. J. Amer. Chem. Soc. <u>77</u>, 2752 (1955).

### CoBr<sub>2</sub> · 6 H<sub>2</sub>O

Precipitated cobalt carbonate is dissolved in aqueous HBr (d 1.49). The solution is heated on the steam bath until a deep blue color appears; it is then concentrated by evaporation on the steam bath. Cooling in ice water precipitates crystals of the hexahydrate; these are filtered off and washed with ice water.

#### PROPERTIES:

Formula weight 326.88. Red crystals. M.p. 47-48°C;  $d_4^{25}$  2.46. Deliquescent in air. All water is removed by standing over conc.  $H_2SO_4$  or by heating to 130-140°C.

**REFERENCE:** 

G. L. Clark and H.K. Bruckner. J. Amer. Chem. Soc. <u>44</u>, 230 (1922).

Cobalt (II) Iodide

## $\alpha$ -CoI<sub>2</sub>, $\beta$ -CoI<sub>2</sub>, CoI<sub>2</sub> · 6 H<sub>2</sub>O

a-CoI2

 $\begin{array}{rcl} \text{Co} & + & 2 \text{ HI} & = & \text{Col}_2 & + & \text{H}_2 \\ \\ 58.9 & & 255.8 & & 312.8 & & 22.4 \ l. \end{array}$ 

Fine Co powder obtained from cobalt oxalate is heated to 400-500 °C in a stream of HI (4-5 hours). The product iodide is melted by heating to 550 °C and allowed to cool in high vacuum.

PROPERTIES:

Black, graphitelike solid. M.p. 515-520 °C (in high vacuum); d<sup>25</sup> 5.584. Heat of formation: -39.13 kcal./mole. Solubility in H<sub>2</sub>O: 58.7% (-2°C); 66.4% (25°C); 80.9% (111°C). Dilute solutions are red; concentrated solutions are red at low temperatures, while at higher temperatures all shades from brown to olive green are present. Very hygroscopic, becomes blackish green in air. Soluble in SOCl<sub>2</sub>, POCl<sub>3</sub>. Crystal structure: C 6 type.

**REFERENCE:** 

W. Biltz and E. Birk. Z. anorg. allg. Chem. 127, 34 (1923).

β-CoI2

Sublimation of  $\alpha$ -CoI<sub>2</sub> in high vacuum yields  $\beta$ -CoI<sub>2</sub>.

The starting  $\alpha$ -CoI<sub>2</sub> is placed at location *a* of the tube shown in Fig. 333 and heated slowly in a high vacuum to 570-575°C. Cobalt metal remains as a residue at *a*, and a black sublimate of  $\alpha$ -CoI<sub>2</sub> is deposited at *b*. Ochre-yellow  $\beta$ -CoI<sub>2</sub> appears at *c*, as do the I<sub>2</sub> crystals present in the tube. The tube is placed in a horizontal furnace in such a way that the section from *a* to *d* is at 100°C. The section projecting from the furnace is fanned with a flame until all I<sub>2</sub> collects at *e*. The loosely adhering  $\beta$ -CoI<sub>2</sub> is shaken from *c* into *f*. The apparatus is filled with N<sub>2</sub>, and *f* with its contents is sealed off. About 0.8 g. of  $\beta$ -CoI<sub>2</sub> is obtained from 10 g. of  $\alpha$ -CoI<sub>2</sub>. Fig. 333. Sublimation of cobalt(II) iodide.

1518

PROPERTIES:

Ochre-yellow powder.  $d_4^{25}$  5.45. Very hygroscopic; deliquesces in moist air, forming green droplets. Solution in H<sub>2</sub>O is colorless, becoming rose-colored on heating. Blackens at 400°C and converts to  $\alpha$ -CoI<sub>2</sub>.

REFERENCE:

E. Birk and W. Biltz. Z. anorg. allg. Chem. 128, 46 (1923).

Col<sub>2</sub> · 6 H<sub>2</sub>O

Precipitated cobalt carbonate or  $Co(OH)_2$  is dissolved in aqueous HI. The solution is concentrated on a steam bath to a sirupy consistency and is then allowed to cool. The product is filtered off and washed with some water.

Alternate method: A solution of  $CoI_2$  in water is cooled to a low temperature [G. L. Clark and H. K. Bruckner, J. Amer. Chem. Soc. <u>44</u>, 230 (1922)].

PROPERTIES:

Formula weight 420.86. d 2.90. Long, dark-red crystals, which begin to lose water of crystallization above 27°C and become anhydrous at 130°C.

**REFERENCES:** 

O. Erdmann. J. prakt. Chem [1] 7, 254 (1836); A. Étard. Ann. Chim. Phys. [7] 2, 503 (1894).

### Cobalt (II) Oxide

#### CoO

Prepared by thermal decomposition of cobalt salts containing a volatile acid moiety.

Cobalt carbonate, precipitated from  $Co(NO_3)_2 \cdot 6H_2O$  with aqueous  $Na_2CO_3$  in the absence of air, is heated for several hours in high vacuum at 350°C.

Analysis for active oxygen is necessary (treatment with hydrochloric acid and determination by the Bunsen method).

Alternate method: (The product is less certain to have the composition CoO): heating of  $Co(NO_3)_2 \cdot 6H_2O$  or  $Co_3O_4$  to 1000°C and cooling in a stream of  $N_2$ .

PROPERTIES:

Formula weight 74.94.  $d_4^{18}$  6.47. Heat of formation: -57.5 kcal./mole. Olive-green powder; takes up O<sub>2</sub> from air at room temperature. Becomes brown, and finally black, as the oxygen content increases. Stable in air when calcined at a high temperature. Converts to Co<sub>3</sub>O<sub>4</sub> on heating in air at 390-900°C. Readily soluble in HCl, H<sub>2</sub>SO<sub>4</sub> and HNO<sub>3</sub>. Fine CoO powder is also soluble in conc. alkali. Crystal structure: B1 (NaCl) type.

**REFERENCE:** 

M. Le Blanc and E. Moebius. Z. phys. Chem. (A) 142, 151 (1929).

# Cobalt (II,III) Oxide

### Co<sub>3</sub>O<sub>4</sub>

Precipitated cobalt carbonate is heated for one hour at 700°C. Analysis for active oxygen is required (treatment with hydrochloric acid and determination by the Bunsen method).

Alternate methods: a) Heating of  $Co(NO_3)_2 \cdot 6H_2O$  at 700°C (crucible, one hour) (J. A. Hedvall and T. Nilson).

b) Heating of CoO in air at 700°C [L. Wöhler and O. Balz, Z. Elektrochem. 27, 406 (1921)].

PROPERTIES:

Formula weight 240.82, d 6.073. Heat of formation: -206.1 kcal./mole. Blackish-gray powder. Converts in air to CoO (905-925°C). Coarse crystalline Co<sub>3</sub>O<sub>4</sub> is attacked only by conc. H<sub>2</sub>SO<sub>4</sub>; fine powder dissolves slowly in acids. Crystal structure: H 11 (spinel) type.

REFERENCE:

J. A. Hedvall and T. Nilson. Z. anorg. allg. Chem. 205, 425 (1932).

# Cobalt (III) Hydroxide

# CoO(OH)

Co(NO <sub>3</sub> ) <sub>2</sub>	+ 3 KOH	$+ 1/_{2} Br_{2} =$	= CoO(OH)	$+ 2 \text{ KNO}_3$	+ KBr $+$	$H_2O$
(6H <sub>2</sub> O) 291.1	168,3	79,9	91.9	202.2	119.0	

A solution of 56 g. of KOH in 300 ml. of  $H_2O$  is added dropwise (stirring) to a solution of 90 g. of  $Co(NO_3)_2 \cdot 6 H_2O$  and 12 ml. of

 $Br_2$  in 1300 ml. of H<sub>2</sub>O. The resulting precipitate settles in about three hours. It is washed by decantation with four 5-liter portions of CO<sub>2</sub>-free water. Should the precipitate undergo peptization on repeated addition of wash water, it is filtered on a Zsigmondy membrane filter, suction-dried and slurried in five liters of CO<sub>2</sub>-free water. The slurry is filtered as above and the solid dried in a vacuum desiccator over conc. H<sub>2</sub>SO<sub>4</sub>.

All operations (precipitation, decantation and filtration) must be conducted in a  $CO_2$ -free atmosphere. After drying in the vacuum desiccator, this precaution is no longer necessary.

Alternate methods: a) Air oxidation of a solution of  $CoCl_2 \cdot 6H_2O$  in aqueous NaOH [W. Feitknecht and W. Bédert, Helv. Chim. Acta <u>24</u>, 683 (1941)].

b) Precipitation of  $[Co(NH_3)_6]Cl_3$  with aqueous KOH [G. F. Huttig and R. Kassler (1929)].

ANALYSES REQUIRED:

Co (electrolytic),  $H_2O$ ,  $CO_2$ , and active oxygen (treatment with hydrochloric acid and determination by the Bunsen method). Used as an oxidation catalyst.

PROPERTIES:

Black powder. d 4.29-4.90. Converts to  $Co_3O_4$  on heating in vacuum at 148-150°C. Dissolves in HCl, evolving  $Cl_2$ . Soluble in HNO<sub>3</sub> and H<sub>2</sub>SO<sub>4</sub>. Not attacked by aqueous alkali or ammonia. Solution in organic acids such as oxalic or tartaric, accompanied by reduction.

REFERENCE:

G. F. Hüttig and R. Kassler. Z. anorg. allg. Chem. <u>184</u>, 279 (1929).

# Cobalt (II) Hydroxide

# Co(OH)2

$Co(NO_3)_2$	+	2 KOH	$= Co(OH)_2$	+	2 KNO <sub>8</sub>
$(6 H_2 O)$					
291.1		112.2	93.0		202.2

ROSE-COLORED Co(OH)2

A solution of 40 g. of  $Co(NO_3)_2 \cdot 6H_2O$  in 1000 g. of  $H_2O$ , cooled to 0°C, is added dropwise and with vigorous shaking to a solution of 40 g. of KOH in 500 g. of  $H_2O$  maintained at 0°C. The

initial blue color of the precipitate rapidly turns to rose. The precipitate is washed by decantation with  $CO_{2}$ - and  $O_{2}$ -free water until K<sup>+</sup> and  $NO_{3}$ - ions can no longer be detected. It is then filtered off and dried in a desiccator over 50% H<sub>2</sub>SO<sub>4</sub>.

Precipitation, washing, filtration and drying must be carried out in an atmosphere free of  $CO_2$  and  $O_2$ , since  $Co(OH)_2$  oxidizes very readily (for technique, see Part I).

ANALYSES REQUIRED:

Co (electrolytic)  $H_2O$  and  $CO_2$ .

**PROPERTIES:** 

Rose-red powder.  $d_4^{15}$  3.597. Heat of formation: --63.4 kcal. per mole. Converts to CoO + H<sub>2</sub>O when heated in vacuum at 168°. Oxidation leads to higher cobalt hydroxides. Readily soluble in acids, insoluble in dil. alkalies, appreciably soluble in ammonia. Crystal structure: C 6 type.

REFERENCE:

G. F. Hüttig and R. Kassler. Z. anorg. allg. Chem. 187, 16 (1930).

BLUE Co(OH)2

A small excess of NaOH solution is added to a 0.1 M cobalt salt solution containing about 1% glucose. The precipitate is thoroughly washed in the absence of air with aqueous alcohol, aqueous acetone and finally with pure acetone. It is then dried. The resulting blue hydroxide changes color very readily on drying (oxidation).

When only small quantities are needed, 5 ml. of the 0°C, 0.1 M cobalt salt solution is treated with 5.2 ml. of 0°C, carbonatefree 0.2 N NaOH in a small Erlenmeyer flask. The precipitate and the mother liquor are poured at once into a centrifuge tube chilled in ice-salt mixture and frozen. The tube, containing the frozen block of solution plus precipitate, is then removed from the ice-salt bath; as soon as the block detaches from the glass tube it is crushed in a porcelain mortar and melted in the centrifuge tube by addition of 25% alcohol. The precipitate is rapidly centrifuged and washed twice with chilled aqueous acetone and with pure acetone. The acetone clinging to the blue precipitate is evaporated in vacuum.

PROPERTIES:

Blue powder. Constitution: "double-layer lattice," related to C 6 type.

1522

REFERENCE:

W. Feitknecht. Helv. Chim. Acta 21, 766 (1938) and private communication.

## **Cobalt Sulfides**

#### CoS, $CoS_2$ , $Co_2S_4$ , $Co_9S_8$

 $\alpha$ -CoS

 $\begin{array}{rcl} \text{Co(NO}_{8})_{2} &+ & \text{H}_{2}\text{S} &= & \text{CoS} &+ & 2 \text{ HNO}_{8} \\ & & & (6\text{H}_{2}\text{O}) \\ & & & 291.1 & & 22.1 \ l. & & 91.0 & & 126.0 \end{array}$ 

Precipitated in the same way as  $\alpha$ -NiS. The product is dried for 90 hours, with the temperature raised slowly from 100 to 540°C. The sulfide dried at 300°C is pyrophoric.

Catalyst for pressure hydrogenation of organic compounds.

PROPERTIES:

Black powder, soluble in HCl. Forms Co(OH)S in air. Amorphous (by x-ray analysis). Heat of formation: -21.71 kcal./mole.

**REFERENCE:** 

E. Dönges, Z. anorg. Chem. 253, 346 (1947).

 $\beta$ -CoS

```
Co + S = CoS
58.9 32.1 91.0
```

Fine Co powder is mixed with the stoichiometric quantity of fine S powder and heated at 650 °C for 2-3 days in an evacuated, sealed quartz tube. The tube is then quenched in cold water.

Analyses for Co and S are necessary. The compound is used as catalyst in the hydrogenation of organic compounds.

Alternate method: A 1 N solution of CoCl<sub>2</sub> is treated with acetic acid and precipitated with  $H_2S$ ; workup is the same as in the case of  $\alpha$ -CoS.

PROPERTIES:

Gray powder. M.p. 1135°C; d 5.45. Soluble in acids. Crystal structure: B8 type. Long heating at 200°C produces a modification with a complicated structure. Material with an overall formula  $\cos_{1.0}$  is not homogeneous; the CoS phase has the composition  $\cos_{1.04}$ -CoS<sub>1.13</sub>.

**REFERENCES:** 

D. Lundquist and A. Westgren. Z. anorg. allg. Chem. <u>239</u>, 85 (1938); E. Dönges. Z. anorg. Chem. 253, 346 (1947).

CoS2

I.

## $Co + 2S = CoS_2$ 58.9 64.1 123.1

Stoichiometric quantities of Co and S powders are mixed and reacted in the same way as described in the case of  $\beta$ -CoS.

### II. REACTION OF H<sub>2</sub>S WITH COBALT (III) COMPLEXES

Dry,  $H_2$ -free  $H_2S$  is allowed to react with  $[Co(NH_3)_5Cl]Cl_2$ or  $[Co(NH_3)_6]Cl_3$ . The temperature is raised to 600-630 °C over a period of one hour, and maintained at this level for two hours. The product is allowed to cool to 200 °C in the stream of  $H_2S$ ; then the  $H_2S$  is displaced with dry  $CO_2$  and the product is cooled further. It is sensitive to air; it is heated with S in a sealed tube (one day at 750 °C), and the excess S is extracted with  $CS_2$  after the reaction is complete.

Analyses for Co and S are required. Used as a catalyst in organic syntheses.

PROPERTIES:

Gray-black crystalline powder; d 4.269. Liberates S when heated in absence of air. Not attacked by nonoxidizing acids or alkalies. Crystal structure: C 2 type.

REFERENCES:

- I. D. Lundquist and A. Westgren. Z. anorg. allg. Chem. 239, 85 (1938).
- II. O. Hülsmann and W. Biltz. Ibid. 224, 73 (1935).

 $Co_3S_4$  (H11 type) and  $Co_3S_8$  (cubic crystalline) are prepared in the same way as  $\beta$ -CoS: heating stoichiometric quantities of Co and S powders in an evacuated, sealed quartz tube at 650°C. [D. Lundquist and A. Westgren (1938)].

# Cobalt (III) Sulfate

### Co<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub> · 18 H<sub>2</sub>O

A sulfuric acid solution of  $CoSO_4 \cdot 7H_2O$  (formula wt. 281.11) is anodically oxidized at 0°C.

1524

A porous clay cell (about 120 ml. capacity) is changed with a solution of 24 g. of  $CoSO_4 \cdot 7H_2O$  in 75 ml. of warm  $8NH_2SO_4$ . A cylindrical Pt sheet (4 cm. high, 12 cm. wide), which serves as anode, is also inserted. A Pt wire welded to the sheet serves as the electrical lead. A Cu cylinder (8 cm. high) with a suitable electrical lead is placed around the clay cell and serves as the cathode. The cathode electrolyte is  $8 NH_2SO_4$ . The electrolysis vessel is cooled in ice water. The electrolysis starts when the anode electrolyte has reached  $30^{\circ}C$  and takes about 12 hours. The thick, deep-blue suspension is rapidly filtered through a fritted-glass funnel and is then pressed dry on a clay plate with a Pt spatula.

Alternate method: Gaseous fluorine is passed through an icecold solution of 24 g. of  $CoSO_4 \cdot 7H_2O$  in 150 ml. of 8 N  $H_2SO_4$ [F. Fichter and H. Wolfmann, Helv. Chim. Acta <u>9</u>, 1093 (1926)].

### PROPERTIES:

Formula weight 730.37. Lustrous, blue-green leaflets. Decomposes rapidly in ice-cold water, liberating  $O_2$  and yielding  $CoSO_4$ . Dilute  $H_2SO_4$  gives a green solution which is stable for several days. Aqueous NaOH precipitates CoO(OH). Decomposes rapidly when heated in dry air, forming a brown (later reddish) powder. Powerful oxidizing agent.

**REFERENCE:** 

E. Muller. Elektrochem. Praktikum [Laboratory Manual for Electrochemistry], 5th ed., Leipzig, 1940, p. 218.

# Cobalt Aluminate

### CoAl<sub>2</sub>O<sub>4</sub>

$$\begin{array}{rcl} \operatorname{CoO} &+ & \operatorname{Al}_2\operatorname{O}_3 &= & \operatorname{CoAl}_2\operatorname{O}_4 \\ 74.9 & & 102.0 & & 176.9 \end{array}$$

A stoichiometric mixture of CoO and  $Al_2O_3$  is prepared and then mixed with 1.5 times its weight of KCl, The mixture is heated to about 1100°C in a porcelain crucible. The melt is cooled, pulverized, and extracted with boiling water until no further Cl<sup>-</sup> reaction is obtained. The residue is dried at 60°C in a drying oven.

SYNONYM:

Thenard's blue.

PROPERTIES:

M.p. 1700-1800°;  $d_4^{18}$  4.37. Not attacked by Cl<sub>2</sub>, mineral acids or aqueous alkalies. Decomposed by fusion with KHSO<sub>4</sub> and by heating with H<sub>2</sub>SO<sub>4</sub> in a sealed tube at 200°C. Crystal structure: H11 (spinel) type.

**REFERENCES**:

J. A. Hedvall. Z. anorg. allg. Chem. 92, 305 (1915); 96, 72 (1916).

### Hexaamminecobalt (III) Nitrate

### [Co(NH<sub>3</sub>)<sub>6</sub>](NO<sub>3</sub>)<sub>3</sub>

 $\begin{bmatrix} Co(NH_3)_6 \end{bmatrix} Cl_8 + 3 HNO_3 = \begin{bmatrix} Co(NH_3)_6 \end{bmatrix} (NO_3)_8 + 3 HCl_{267.5} \\ 189.0 & 347.2 \\ 109.4 \end{bmatrix}$ 

A solution of  $[Co(NH_3)_6]Cl_3$  in a minimum quantity of water is prepared and dil. HNO<sub>3</sub> is added; the resulting precipitate is washed with dil. HNO<sub>3</sub> until free of the chloride ion, then with 90% alcohol until free of acid.

PROPERTIES:

Yellow tetragonal prisms.  $d_4^{25}$  1.804. Solubility in H<sub>2</sub>O in moles/liter: 0.0202 (0°C); 0.052 (20°C); 0.0704 (30°C).

**REFERENCE:** 

S. M. Jörgensen, Z. anorg. Chem. <u>17</u>, 457 (1898).

### Cobalt (III) Amide

#### Co(NH<sub>2</sub>)<sub>3</sub>

Fluffy Co(NH<sub>2</sub>)<sub>3</sub> precipitates upon addition of KNH<sub>2</sub> to a solution of  $[Co(NH_3)_6]$  (NO<sub>3</sub>)<sub>3</sub> in liquid NH<sub>3</sub>.

 $[Co(NH_3)_6](NO_3)_8 + 3 KNH_2 = Co(NH_2)_8 + 6 NH_3 + 3 KNO_3$ 347.2 165.4 107.0 102.2 303.3

The process requires a supply of  $N_2$  entirely free of oxygen traces (see p. 458 ff.); this nitrogen is then passed through a wash bottle with conc. KOH, a drying tower with solid KOH, and two U tubes with  $P_2O_5$ . The stream of  $N_2$  is then led to a manifold,

from which it may be directed through stopcocks to various reaction vessels and auxiliary apparatus such astransfer and storage tubes. A branch terminating in a mercury-filled beaker serves as a pressure-relief valve.

In addition, a stream of very pure NH<sub>3</sub> is required; this may be taken from a storage cylinder which contains some Na metal. The NH<sub>3</sub> is passed through a drying tube containing NaNH2 and then a frittedglass disk (not too fine) to remove any entrained solid particles. The NH<sub>3</sub> line has two side branches, one terminating in a vent stopcock, the other in an Hg manometer. The NH3 line must be capable of withstanding pressures to about 10 atm.; thus, it must contain no ground joints and only a few well-secured stopcocks. The NH<sub>3</sub> line itself terminates in the above-mentioned manifold.

The reaction takes place in the pressure-resistant vessel shown in Fig. 334. This vessel and accessory equipment are attached to the manifold via flexible couplings made of lead tubing or corrugated pinchbeck (copper-zinc alloy); thus, they may be connected to the  $N_2$  or  $NH_3$  streams, as desired, and also can be moved to some extent.

a c no cm b

Fig. 334. Pressure vessel for preparation of cobalt (III) amide. Stopcock h and associated tubing are perpendicular to the plane of the diagram. The plug in his held in place by a clamp to prevent it from being blown out by pressure in the apparatus.

The pressure apparatus is first heated while the N<sub>2</sub> stream passes through. Then, 1.5-2 g. of [Co(NH<sub>3</sub>)<sub>6</sub>](NO<sub>3</sub>)<sub>3</sub> is admitted through a from a charging funnel and placed on the fritted-glass disk. The tube is then sealed at c. Next, a few mg. of Pt black and the required quantity of metallic K (3 moles of K per mole of  $[Co(NH_3)_6](NO_3)_3 + 5\%$  excess K) are placed in reactor b, and the tube is sealed at d. The stopcock on the N<sub>2</sub> line is closed and the valve on the NH<sub>3</sub> cylinder is opened. When the manometer in the NH<sub>3</sub> line shows about 4 atm., the valve is closed and the NH<sub>3</sub> is vented to the atmosphere by opening the vent stopcock. This purging process is repeated three times to displace the N<sub>2</sub> from the apparatus. Then reactor b is immersed in ice-salt mixture, the valve on the NH<sub>3</sub> cylinder fully opened, and NH<sub>3</sub> allowed to condense in b until the latter is 3/4 full of liquid. Stopcock h is then closed and tube a is cooled so that NH<sub>3</sub> distills from b into a. The hydrogen evolved in the reaction of K with NH<sub>3</sub> is vented

from time to time by carefully opening h and the vent stopcock in the NH<sub>3</sub> line. All of the potassium is allowed to react, whereby all the  $[Co(NH_3)_6](NO_3)_3$  dissolves. The apparatus is then tilted while tube a is cooled. The KNH<sub>2</sub> solution is thereby filtered into a (at the same time, the apparatus is rocked to mix the contents). Reactor b is then recooled, thus recondensing the NH<sub>3</sub> in it; this liquid  $NH_3$  is again filtered into *a*. This procedure is repeated three times in order to react all of the KNH<sub>2</sub>. The reaction mixture is allowed to remain in a for six hours (ice cooling and frequent shaking). It is then filtered into b, the precipitate being retained on the fritted-glass disk. The ammonia is again evaporated from b and condensed in a, shaken there with the solid, and filtered into b once more. This is repeated ten times, after which the precipitate is completely free of KNO<sub>3</sub>. Finally, the NH<sub>3</sub> is vented by opening h as well as the vent stopcock in the NH<sub>3</sub> line. Nitrogen is then introduced into the apparatus, which is then inverted and opened by breaking the seal at c. The amide on the filter plate is crushed with the aid of a bent Ni spatula, and transferred (in a stream of  $N_2$ ) through the open end of *a* into a storage device (see Part I, Fig. 54b for the latter). This storage tube is also purged with very pure N2 and, after the amide has been introduced, is closed with a ground stopper. The storage tube is cooled in ice while  $N_2$  is passed over the amide until no further  $NH_3$  is given off.

ANALYSIS:

The product amide is placed in a small glass bulb (air must be excluded). The bulb is then sealed. The following may then be done:

Determination of  $NH_3$ : Aqueous KOH is added to the amide and the liberated  $NH_3$  is distilled into a known quantity of acid.

Determination of Co: The Co(OH)<sub>3</sub> which precipitates on addition of KOH is dissolved in sulfurous acid and the Co determined by analytical electrolysis.

The  $NO_3$  is determined with diphenylamine.

PROPERTIES:

Chocolate-brown powder; sometimes pyrophoric. Soluble indil. acids with brown-red color, and in cold water with brown color; the slightly cloudy liquid gradually deposits brownish-black cobalt (III) hydroxide. Forms CoN in vacuum at  $40-50^{\circ}$ C. Heating in liquid NH<sub>3</sub> produces CoN, Co<sub>2</sub>N and CoN<sub>0.28</sub>.

REFERENCE:

O. Schmitz-Dumont, J. Pilzecker and H. F. Piepenbrinck. Z. anorg. allg. Chem. <u>248</u>, 175 (1941).

# **Dicobalt Nitride**

### Co<sub>2</sub>N

Ten mg. of  $Co_3O_4$  is reduced with pure  $H_2$  by heating in a corundum boat for two hours at 350°C. The resulting Co powder is then heated for three hours at 380°C in a stream of  $NH_3$  which passes through the tube at 22 cm./sec. The product is ground and treated once more under the same conditions.

The reduction and nitridation must be carried out in one continuous operation, since the Co powder obtained by reduction of  $Co_3O_4$  is pyrophoric. The cobalt powder prepared from cobalt oxalate cannot be completely converted to nitride under these conditions.

Alternate method: Thermal degradation of  $Co(NH_2)_3$  in vacuum at 160°C [O. Schmitz-Dumont, Angew. Chem. <u>67</u>, 231 (1955); J. Clarke and K. H. Jack, Chem. and Ind. <u>1951</u>, 1004].

#### PROPERTIES:

Grayish-black powder. d 6.4. In the cold, dil. acids and alkalies react slowly, conc. HCl rapidly, and conc. HNO<sub>3</sub> violently. Warm dil. acids also dissolve  $Co_2N$  rapidly. The slow attack by acids gives a quantitative yield of the NH<sub>4</sub>salts (this is an analytical method), while vigorous decomposition evolves part of the nitrogen as N<sub>2</sub>. Forms a nitride with the approximate composition  $Co_3N$  on thermal decomposition. In vacuum, stable until formation of the compound  $CoN_{0.41}$  at 200°C; at 250°C, hexagonal metallic Co containing small amounts of Ni is produced. Crystal structure: rhombically deformed hexagonal close packing of metal atoms.

**REFERENCE:** 

R. Juza and W. Sachse. Z. anorg. Chem. 253, 45 (1945).

### **Cobalt Nitride**

#### CoN

 $Co(NH_2)_3 = CoN + 2 NH_3$ 107.0 72.9 44.2 l.

Cobalt amide is placed in a vapor-pressure eudiometer (see Part I, Fig. 85) and carefully decomposed at 50-70°C in the absence

of air. The NH<sub>3</sub> evolved is absorbed in conc.  $H_2SO_4$ . The degradation is continued until all NH<sub>3</sub> has been eliminated. The CoN thus formed is transferred in the absence of air to a glass bulb, which is then sealed off. The exact stoichiometric composition is never attained, since a small amount of N<sub>2</sub> is evolved along with the NH<sub>3</sub>. The degration products usually have the composition  $CoN_{0.8}$ - $CoN_{0.9}$ .

### ANALYSIS:

To determine the valence of Co, the sample is carefully heated with 2 N KOH until  $NH_3$  can no longer be detected. The resulting blue liquid, which contains suspended cobalt (III) hydroxide, is treated with KI and  $NaHCO_3$  in a flask closed off with a glass stopper, and is then carefully acidified. After standing for one day, the  $I_2$  which separates is back-titrated with  $Na_2S_2O_3$  solution.

### PROPERTIES:

Black powder; pyrophoric. The nitrogen content of the degradation product drops off with increasing degradation temperature (the composition  $CoN_{0.5}$ , corresponding to the formula  $Co_2N$ , is obtained at 160°C). Heating in the presence of H<sub>2</sub>O and aqueous alkali liberates NH<sub>3</sub>. Dilute H<sub>2</sub>SO<sub>4</sub> liberates part of the bound nitrogen as N<sub>2</sub>. Crystal structure: B1 type.

REFERENCES:

O. Schmitz-Dumont, H Broja and H. F. Piepenbrink. Z. anorg. Chem. <u>253</u>, 118 (1947); O. Schmitz-Dumont. Angew. Chem. <u>67</u>, 231 (1955).

# **Cobalt Phosphides**

### CoP<sub>3</sub>, CoP, Co<sub>2</sub>P

Prepared by heating stoichiometric quantities of pure Co metal and red P for 20 hours at 650-700°C in sealed, evacuated quartz tubes. The starting Co powder is obtained by reduction of CoO or  $Co_3O_4$  with H<sub>2</sub> at 700°C.

 $C_0 + 3P = C_0P_3$ ;  $C_0 + P = C_0P$ ;  $2C_0 + P = C_{02}P$ 

58.9 92.9 151.9 58.9 31.0 89.9 117.9 31.0 148.9

PROPERTIES:

- CoP<sub>3</sub>: Grayish-black powder; d<sub>4</sub><sup>25</sup> 4.26.
- CoP: Grayish-black powder; d<sup>25</sup> 6.24. Crystal structure: B 21 type.
- Co<sub>2</sub>P: Grayish-black powder; d<sup>25</sup><sub>4</sub> 7.4. Crystal structure: C 23 type.

**REFERENCES:** 

- CoP<sub>3</sub>: W. Biltz and M. Heimbrecht. Z. anorg. allg. Chem. <u>241</u>, 349 (1939).
- CoP: K. E. Fylking. Ark. Kem. Mineral. Geol. <u>11</u> (B), No. 48 (1934).
- Co<sub>2</sub>P: H. Nowotny. Z. anorg. Chem. <u>254</u>, 31 (1947).

## Dicobalt Carbide

### Co<sub>2</sub>C

Fine Co powder, obtained from CoO and  $H_2$  at 280-300°C, is heated at 220°C with dry,  $O_2$ -free CO (flow rate 0.75 liter/hour) for 550 hours. Reduction of the oxide and preparation of the carbide must be carried out in one continuous operation, since the Co powder is pyrophoric. The temperature must be held exactly at 220°C since  $Co_2C$  decomposes above 225°C.

ANALYSIS:

Heating of  $\text{Co}_{2}$ C (9.24% C) with H<sub>2</sub> at 250°C gives a quantitative yield of CH<sub>4</sub> (free C gives no CH<sub>4</sub> under these conditions).

PROPERTIES:

Metallic-gray powder. Decomposes between 260 and 310 °C. Hydrogen converts the orthorhombic carbide to a hexagonal form between 198 and 275 °C, N<sub>2</sub> between 297 and 369 °C, and CO<sub>2</sub> between 364 and 540 °C. Space group of the rhombic carbide:  $d_{2h}^{12}$ .

**REFERENCES:** 

H. A. Bahr and V. Jessen. Ber. dtsch. chem. Ges. <u>63</u>, 2226 (1930);
 J. E. Hofer and W. C. Beebles. J. Amer. Chem. Soc. <u>69</u>, 893 (1947).

### Hexaamminecobalt (III) Chloride

## [Co(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>3</sub>

$4 \operatorname{CoCl}_2 +$	-4 NH <sub>4</sub> Cl $-$	$+ 20 \text{ NH}_{3}$	$+ O_2 =$	$4 [Co(NH_3)_6]Cl_3$	$+ 2 H_2$	0
(6H <sub>2</sub> O) 951.8	214.0	340.6	22.4 l.	1070.0	36.0	)

A mixture of 240 g. of  $CoCl_2 \cdot 6H_2O$ , 160 g. of  $NH_4Cl$ , and 200 ml. of  $H_2O$  is shaken until solution is almost complete. Then 4-5 g.

of activated charcoal and 500 ml. of conc. ammonia are added, and a fast stream of air is passed through the mixture until the red solution becomes yellow-brown. The air flow should not be so rapid as to reduce the ammonia content; should this occur, some additional conc. ammonia may be added.

The precipitated  $[Co(NH_3)_6]Cl_3$  and the charcoal are filtered off, and the residue is dissolved in hot 1-2% HCl. The solution is filtered hot and the pure product is precipitated by adding 400 ml. of conc. HCl and chilling to 0°C. The precipitate is collected, washed first with 60% alcohol, then with 95% alcohol, and finally dried at 80-100°C. Yield: 85%.

Alternate method: From  $[Co(NH_3)_5Cl]Cl_2$  and ammonia [S. M. Jörgensen, Z. anorg. Chem. <u>17</u>, 455 (1898)].

SYNONYM:

Luteocobalt chloride.

PROPERTIES:

Wine-red or brownish-red monoclinic crystals.  $d_4^{25}$  1.710. Solubility in H<sub>2</sub>O in moles/liter: 0.152 (0°C); 0.26 (20°C); 0.42 (46.6°C); Boiling in water yields Co(OH)<sub>2</sub>.

REFERENCES:

J. Bjerrum. Metal Ammine Formation in Aqueous Solution, p. 241, Copenhagen, 1941; J. Bjerrum and J. P. McReynolds in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 217.

# Chloropentaamminecobalt (III) Chloride

# [Co(NH<sub>3</sub>)<sub>5</sub>Cl]Cl<sub>2</sub>

Obtained by oxidation of an ammoniacal CoCl<sub>2</sub> solution, and purified via  $[Co(NH_3)_5H_2O]_2(C_2O_4)_3 \cdot 4H_2O$ .

A) CRUDE PRODUCT, [Co(NH<sub>3</sub>)<sub>5</sub>Cl]Cl<sub>2</sub>

A solution of 20 g. of precipitated cobalt carbonate in some 1:1 HCl is prepared, filtered and cooled; a mixture of 250 ml. of conc. ammonia and 50 g. of  $(NH_4)_2CO_3$  dissolved in 250 ml. of  $H_2O$  is then added. The mixture is oxidized for three hours by bubbling in a stream of air. After addition of 150 g. of  $NH_4Cl$  the solution is evaporated to sirup consistency on the steam bath. Dilute HCl is added to drive off the  $CO_2$  and produce a weakly acid reaction;

then ammonia is added to give a weakly basic solution, followed by 10 ml. of additional conc. ammonia. The liquid, whose volume at this point is 400-500 ml., is heated on the steam bath until all the tetraammine salt disappears; it is then treated with 300 ml. of conc. HCl and heated for 30-45 minutes on the steam bath. The  $[Co(NH_3)_5-Cl]Cl_2$  precipitates on cooling. It is filtered off and washed with 1:1 HCl until free of NH<sub>4</sub>Cl, then with alcohol until free of acid. The salt still contains some  $[Co(NH_3)_5]Cl_2$ . Yield: 34.5 g.

**REFERENCE:** 

S. M. Jörgensen. Z. anorg. Chem. 5, 361 (1894).

B) AQUOPENTAAMMINECOBALT (III) OXALATE,  $[C_0(NH_3)_5(H_2O)]_2 (C_2O_4)_3 \cdot 4H_2O$ 

A mixture of 10 g. of finely powdered crude  $[Co(NH_3)_5Cl]Cl_2$ , 75 ml. of  $H_2O$ , and 50 ml. of 10% ammonia is heated on the steam bath in an Erlenmeyer flask covered with a watch glass (continuous agitation) until all of the basic aquopentaammine chloride dissolves and a deep-red solution forms. The solution is filtered, the filtrate is made very weakly acid with oxalic acid, and some additional  $(NH_4)_2C_2O_4$  is added to complete the precipitation. The slurry is allowed to stand; the precipitate is then filtered off and washed with cold water. The yield of the dry salt is about 12 g.

SYNONYM:

Roseocobalt oxalate.

PROPERTIES:

Formula weight 660.36. Brick-red crystals. Solubility in water at 17.5°C is 0.0019 moles/liter.

REFERENCE:

S. M. Jörgensen. Z. anorg. Chem. 19, 78 (1899).

C) PURE CHLOROPENTAAMMINECOBALT (III) CHLORIDE, [Co(NH<sub>3</sub>)<sub>5</sub>Cl]Cl<sub>2</sub>

Twenty grams of  $[Co(NH_3)_5(H_2O)]_2(C_2O_4)_3 \cdot 4H_2O$  is dissolved in 150 ml. of 2% ammonia in the cold, and the insoluble  $[Co(NH_3)_6]_2$ - $(C_2O_4)_3 \cdot 4H_2O$  (luteooxalate) is filtered off. The filtrate is precipitated in the cold with dil. HCl. The  $[Co(NH_3)_5Cl]Cl_2$  thus formed is filtered off, washed successively with alcohol, absolute alcohol and ether, and dried in air. The purification method given by Jörgensen does not yield completely pure  $[Co(NH_3)_5Cl]Cl_2$ .

ANALYSIS:

Ionizable Cl and total Cl are determined in order to determine whether impurities are present.

SYNONYM:

Chloropurpureocobalt chloride.

PROPERTIES:

Formula weight 250.47. Ruby-red crystals.  $d_4^{25}$  1.783. Solubility in H<sub>2</sub>O: 0.0089 (0°C); 0.0225 (25°C); 0.040 (50°C) moles/liter. The presence of HCl lowers the solubility; at 25°, 10% HCl disolves 0.00067 moles/liter. Neutral aqueous solutions decompose when boiled, and Co(OH)<sub>2</sub> is deposited. Heating to higher temperatures produces CoCl<sub>2</sub>. Crystal type: orthorhombic-bipyramidal.

REFERENCE:

F. J. Garrick. Z. anorg. allg. Chem. 224, 27 (1935).

Nitropentaamminecobalt (III) Chloride

# $[Co(NH_3)_5NO_2]Cl_2$

 $[Co(NH_3)_5Cl]Cl_2 + NaNO_2 = [Co(NH_3)_5NO_2]Cl_2 + NaCl_{250.5} 69.0 261.0 58.4$ 

A mixture of 20 g. of  $[Co(NH_3)_5C1]Cl_2$ , 200 ml. of  $H_2O$ , and 50 ml. of 10% ammonia is placed in an Erlenmeyer flask covered with a watch glass, and heated on the steam bath until the salt dissolves (frequent shaking is necessary). The solution is filtered, the filtrate cooled and made weakly acidic with dil. HCl, 25 g. of crystalline NaNO<sub>2</sub> is added, and heating on the steam bath is continued until the initial red precipitate dissolves completely. The cold, brownish-yellow solution contains a copious deposit of crystals. At this point, 250 ml. of conc. HCl is added (carefully at first). After chilling, the product is filtered off, washed with 1:1 HCl, then with alcohol until free of acid, and dried in air. Yield: 17 g.

SYNONYM:

Xanthocobalt chloride.

PROPERTIES:

Brownish-yellow monoclinic crystals.  $d_4^{25}$  1.804. Solubility in H<sub>2</sub>O at 20°C: 0.11 moles/liter of solution; more soluble in hot H<sub>2</sub>O. Decomposes on heating to 210°C.

**REFERENCE:** 

S. M. Jörgensen, Z. anorg. Chem. <u>17</u>, 463 (1898).

# Nitritopentaamminecobalt (III) Chloride

#### [Co(NH<sub>3</sub>)<sub>5</sub>ONO]Cl<sub>2</sub>

 $\begin{bmatrix} Co(NH_3)_5Cl \end{bmatrix}Cl_2 + NaNO_2 = \begin{bmatrix} Co(NH_3)_5ONO \end{bmatrix}Cl_2 + NaCl_{250,5} & 69.0 & 261.0 & 58.4 \end{bmatrix}$ 

A solution of 10 g. of  $[Co(NH_3)_5Cl]Cl_2$  in a mixture of 150 ml. of H<sub>2</sub>O and 25 ml. of 10% ammonia is prepared with heating and agitation. The solution is filtered, cooled and exactly neutralized with dil. HCl. Then 25 g. of crystalline NaNo<sub>2</sub> is added and, when this has dissolved, 10 ml. of 1:1 HCl. The resulting precipitate is allowed to stand for several hours in the mother liquor while cooling in water; it is then filtered off and washed with cold water and alcohol.

PROPERTIES:

Chamois-colored crystalline powder, four times less soluble in water than nitropentaamminecobalt chloride. On standing for several weeks, converts to the isomeric form. The conversion is more rapid if a 10% aqueous solution of the compound is treated with an equal volume of conc. HCl.

**REFERENCE**:

S. M. Jörgensen, Z. anorg. Chem. 5, 168 (1894).

### Carbonatotetraamminecobalt (III) Sulfate

#### $[Co(NH_3)_4CO_3]_2SO_4 \cdot 3H_2O$

An aqueous solution of  $CoSO_4$  is treated with  $(NH_4)_2CO_3$  and  $NH_4OH$ , then oxidized in a stream of air.

Precipitated cobalt carbonate (20 g.) is dissolved in a minimum quantity of dil.  $H_2SO_4$ . The clear solution (about 100 ml.) is poured

into a solution of 100 g. of  $(NH_4)_2CO_3$  in 500 ml. of H<sub>2</sub>O and 250 ml. of conc. ammonia, and air is bubbled through for 2-3 hours. After complete oxidation, the blood-red solution, containing several small pieces of  $(NH_4)_2CO_3$ , is evaporated on a steam bath until the volume is 300 ml. The solution is then filtered, concentrated to 200 ml. and chilled, whereupon  $[Co(NH_3)_4CO_3]_2SO_4 \cdot 3H_2O$  crystallizes as red prisms. The mother liquor is decanted; the precipitate is filtered off and washed once with a saturated solution prepared from a small portion of the precipitate. More salt is obtained by further evaporation of the mother liquor [add some  $(NH_4)_2CO_3$ ]. Yield: 32 g.

PROPERTIES:

Formula weight 524.27. Garnet-red prisms. d 1.882. The aqueous solution decomposes on standing in light. Forms  $[Co(NH_3)_4 - (H_2O)_2]SO_4$  (tetraammineroseocobalt sulfate) with dil.  $H_2SO_4$ . Loses all water over conc.  $H_2SO_4$ . Crystal form: orthorhomic-bipyramidal.

**REFERENCE**:

S. M. Jörgensen, Z. anorg. Chem. 2, 281 (1892).

Dichlorotetraamminecobalt (III) Chloride

# [Co(NH<sub>3</sub>)<sub>4</sub>Cl<sub>2</sub>]Cl

Two stable isomeric forms exist: these are the 1,2-dichloro-(cis-) and 1,6-dichloro- (trans-) compounds.

1,2-DICHLOROTETRAAMMINECOBALT (III) CHLORIDE (CIS),  $[Co(NH_3)_4Cl_2]Cl + 0.5H_2O$ 

Treatment of an ammoniacal solution of  $Co(CH_3COO)_2 \cdot 4H_2O$ with NaNO<sub>2</sub>, followed by air oxidation, affords  $[Co(NH_3)_4(NO_2)_2]$ -NO<sub>2</sub>, which is converted to the dichloro compound with conc. HCl.

Air is bubbled for five hours through a solution of 20 g. of NaNO<sub>2</sub> and 20 g. of Co(CH<sub>3</sub>COO)<sub>2</sub>  $\cdot$  4 H<sub>2</sub>O in 200 ml. of 20% NH<sub>3</sub>. The violet solution is concentrated to a small volume with occasional addition of solid NaHCO<sub>3</sub>, then chilled. A large excess of alcohol is added to cause precipitation. The precipitate is filtered off, washed with alcohol-ether, and dried in a vacuum desiccator. This is *cis*-dinitrotetraamminecobalt (III) nitrite, which is not contaminated with the *trans* compound. It is very easily hydrolyzed.

The cis-nitrite is added in small portions to conc. HCl maintained at  $-10^{\circ}$ C, giving a quantitative yield of the chloride. SYNONYM:

Formerly: Chlorovioleocobalt chloride.

PROPERTIES:

Formula weight 242.45. Violet needles. Water soluble; loses water of hydration at 60°C. Very unstable.

REFERENCE:

C. Duval. Comptes Rendus Hebd. Séances Acad. Sci. 182, 636 (1926).

DIAQUOTETRAAMMINECOBALT (III) SULFATE,  $[C_0(NH_3)_4(H_2O)_2]_2(SO_4)_3 \cdot 3H_2O$ 

This is formed by treatment of  $[Co(NH_3)_4CO_3]_2(SO_4)_3 \cdot 3H_2O$  with dil.  $H_2SO_4$ .

A solution of 5 g. of pure  $[Co(NH_3)_4CO_3]_2SO_4 \cdot 3H_2O$  in 100 ml. of cold  $H_2O$  and 10 ml. of dil.  $H_2SO_4$  is prepared; this results in evolution of  $CO_2$ . The clear solution is treated with 50-60 ml. of alcohol, added in small portions. The precipitate is filtered off, washed with 50% alcohol until free of acid, and dried in air. Yield: 6.2 g. (theoretical: 6.37 g.).

PROPERTIES:

Formula weight 668.45. Red quadratic prisms, which lose water of crystallization over conc.  $H_2SO_4$ . Solubility at 22 °C: 0.175 moles/liter of water.

REFERENCE:

S. M. Jörgensen, Z. anorg. Chem. 2, 296 (1892).

1,6-DICHLOROTETRAAMMINECOBALT (III) CHLORIDE (TRANS), [Co(NH<sub>3</sub>)<sub>4</sub>Cl<sub>2</sub>]Cl · H<sub>2</sub>O

A solution of 10 g. of  $[Co(NH_3)_4(H_2O)_2]_2(SO_4)_3 \cdot 3H_2O$  in 50 ml. of cold conc.  $H_2SO_4$  is prepared; the flask is allowed to stand for 24 hours, then placed in ice, and 50 ml. of conc. HCl is added dropwise with frequent and vigorous shaking. The *trans* salt separates in 48 hours. The flask is tilted and the mother liquor decanted as thoroughly as possible. Dilute HCl is then added; the precipitate is filtered off and washed with dil. HCl until free of  $H_2SO_4$ , then washed with alcohol until free of acid. Yield: 7.25 g. (theoretical: 7.53 g.).

SYNONYM:

Formerly: Chloropraseocobalt chloride.
PROPERTIES:

Formula weight 251.46. Lustrous green crystals.  $d^{25}$  1.860. Loses water of crystallization in 1-2 hours at 100°C. Solubility at 0°: 0.0141 moles/liter H<sub>2</sub>O; hydration in solution, yielding  $[Co(NH_3)_4(H_2O)_2]Cl_3$ . The *trans* compound is more stable than the *cis* form.

REFERENCE:

S. M. Jörgensen. Z. anorg. Chem. <u>14</u>, 404 (1897).

# Triethylenediaminecobalt (III) Bromide

# [Co en<sub>3</sub>]Br<sub>3</sub>

The preparation from cobalt salt, ethylenediamine and NaBr yields the racemate of the optically active forms of  $[Co en_3]Br_3$  as the first product. The racemate can be resolved with tartaric acid into the d- and l-tartrate, and further converted to the d- and l-bromide.

A) RACEMIC TRIETHYLENEDIAMINECOBALT (III) BROMIDE,  $[Co en_3]Br_3 \cdot 3H_2O$ 

A solution of 10 g. of  $CoCl_2 \cdot 6 H_2O$  in 150 g. of 10% aqueous ethylenediamine is prepared and oxidized by bubbling air through it for several hours. The brown solution is then acidified with HCl and concentrated until crystallization. The crystal mass is dissolved in  $H_2O$  and treated with  $NH_4NO_3$ , which precipitates 1,6-[Co  $en_2Cl_2$ ]NO<sub>3</sub>. This precipitate is removed by filtration; then NaBr is added to the filtrate, whereupon completely pure [Co  $en_3$ ]Br<sub>3</sub> · 3 H<sub>2</sub>O separates out.

PROPERTIES:

Formula weight 533.04. Small yellow needles. M.p. 271°;  $d_4^{25}$  1.845. Solubility in H<sub>2</sub>O at 16°C: 4.33 g. of anhydrous salt per 100 g. of solution.

## B) RESOLUTION WITH TARTARIC ACID

A solution of 100 g. of  $[Co en_3]Br_3$  in water is treated with the amount of silver tartrate (68.3 g.) needed for reaction with two atoms of bromine. After boiling, the AgBr precipitate is filtered off and then washed with boiling water until the water is no longer yellow. The filtrate and washings are combined and concentrated. On chilling, the d-tartrate separates and is removed by filtration. The mother liquor is further concentrated and the additional precipitate of d-tartrate is removed. Cooling gels the solution to a mass of l-tartrate, still somewhat contaminated with d-tartrate.

# C) d-TRIETHYLENEDIAMINECOBALT (III) BROMIDE TARTRATE $[Co\ en_3]Br(d-C_4H_4O_6) \cdot 5H_2O$

The *d*-tartrate crystals obtained in B are recrystallized from hot water. Rapid cooling yields needles with a silky luster; these disappear on standing for 1-2 hours in the mother liquor and are replaced by coarse platelike crystals.

## PROPERTIES:

Formula weight 557.32. Small bright-yellow needles or darkyellow platelets. Optical rotation (1% solution)  $[\alpha]_D$  +98°, [M]\_D +555°.

# D) *d*-TRIETHYLENEDIAMINECOBALT (III) BROMIDE, d-[Co en 3]Br3 · 2H<sub>2</sub>O

The d-[Co en<sub>3</sub>]Br(C<sub>4</sub>H<sub>4</sub>O<sub>6</sub>)  $\cdot$  5 H<sub>2</sub>O obtained above is triturated with warm conc. HBr and the solution is filtered. On standing, large hexagonal tablets (acid bromide?) separate out; these are recrystallized from water to yield columnar crystals of the dbromide.

## PROPERTIES:

Formula weight 515.03. Yellow, columnar crystals. More readily soluble in H<sub>2</sub>O than the racemic bromide.  $d_4^{25}$  1.971. Optical rotation (1% solution):  $[\alpha] D + 117^{\circ}$ ;  $[M] D + 602^{\circ}$ . Crystal-line form: ditetragonal-bipyramidal.

E) *l*-TRIETHYLENEDIAMINECOBALT (III) BROMIDE, *l*-[Co en<sub>3</sub>]Br<sub>3</sub> ·  $2H_2O$ 

The gelatinous *l*-bromide tartrate is triturated with warm conc. HBr. The sparingly soluble racemic bromide tartrate which separates out is removed by filtration. On standing, the solution deposits crystals of *l*-bromide, which are recrystallized from hot water.

## PROPERTIES:

Formula weight 515.03. Yellowish crystals. More readily soluble in  $H_2O$  than the racemic bromide. Optical rotation (1% solution):  $[\alpha]_D - 115^\circ$ ;  $[M]_D - 592^\circ$ . Can be converted (with AgCl or AgNO<sub>3</sub>) to the corresponding chloride or nitrate.

REFERENCE:

# A-E: A. Werner. Ber. dtsch. chem. Ges. 45, 121 (1912).

Decaammine-µ-peroxocobalt (III) Cobalt (IV) Sulfate

 $[(NH_3)_5Co^{111}(O_2)Co^{1V}(NH_3)_5](SO_4)_2 \cdot SO_4H \cdot 3H_2O$ 

The preparation involves oxidation of an ammoniacal solution of  $CoSO_4 + (NH_4)_2SO_4$ .

A mixture of 0.5 liter of 1 M  $(NH_4)_2SO_4$ , 1 liter of conc. ammonia, 1 liter of H<sub>2</sub>O, 0.5 liter of 1 M CoSO 4, 0.5 liter of 1 M H<sub>2</sub>O<sub>2</sub> and 0.5 liter of 1 M  $(NH_4)_2S_2O_8$  is prepared in the indicated sequence, the solutions being added at approximately 10-second intervals. The mixture is vigorously agitated after each addition. After all additions have been completed it is allowed to stand for 10-15 minutes. Most of the supernatant liquid is siphoned off; the precipitate is filtered off, washed first with dil. ammonia and then with alcohol and suction-dried. The crude product (50-70 g.) is dissolved as rapidly as possible in 1250-1750 g. of 2 N H<sub>2</sub>SO<sub>4</sub> (heating to 80-85°C is necessary). The solution is filtered at once and allowed to stand one day. The yield is 30-50 g. of pure product.

ANALYSIS:

 $2 \left[ Co_{2}(NH_{3})_{10}O_{2} \right]^{5+} + 20 H^{+} + As^{3+} = 2 O_{2} + 4 Co^{2+} + 20 NH_{4}^{+} As^{5+}.$ 

The product, in the sulfuric acid solution, is reduced with  $As^{3+}$ ; the evolving  $O_2$  is collected in an azotometer over strong KOH by means of  $CO_2$ . An aliquot of the solution is used to backtitrate the excess  $As^{3+}$  with  $Ce(SO_4)_2$  and ferroin; another aliquot is used to determine Co by precipitation with 8-hydroxyquinoline and titration with KBrO<sub>3</sub>.

PROPERTIES:

Formula weight 663.45. Grayish-black, lustrous prisms. Almost insoluble in cold dil. sulfuric acid; more soluble at 50-60°C.

REFERENCE:

K. Gleu and K. Rehm. Z. anorg. allg. Chem. 237, 79 (1938).

# Sodium Hexanitritocobaltate (III) Nas[Co(NO2)6]

$$2 \operatorname{Co}(\operatorname{NO}_3)_2 \cdot 6 \operatorname{H}_2 O + 12 \operatorname{Na} \operatorname{NO}_2 + 2 \operatorname{CH}_3 \operatorname{COOH} + \frac{1}{2} \operatorname{O}_2$$
  
582.1 828.0 120.1 11.2 l.  

$$= 4 \operatorname{Na} \operatorname{NO}_3 + 2 \operatorname{CH}_3 \operatorname{COONa} + 2 \operatorname{Na}_3 \operatorname{Co}(\operatorname{NO}_2)_6 + 7 \operatorname{H}_2 O$$
  
340.0 164.1 807.9

A solution of 150 g. of NaNO<sub>2</sub> in 150 ml. of H<sub>2</sub>O is cooled to 50-60°C; some of the NaNO<sub>2</sub> is thus reprecipitated. Then 50 g. of Co(NO<sub>3</sub>)<sub>2</sub> · 6 H<sub>2</sub>O is added, followed by 50 ml. of 50% CH<sub>3</sub>COOH in small portions (agitation). Then a fast stream of air is bubbled through for one half hour. After standing for two hours, the brown precipitate is filtered off. The filtrate must be clear at this point. The precipitate is stirred with 50 ml. of H<sub>2</sub>O at 70-80°C. The solution is separated from undissolved K<sub>3</sub>[Co(NO<sub>2</sub>)<sub>8</sub>] on a small filter and combined with the above-mentioned clear filtrate. The combined solution (about 300 ml.) is treated with 250 ml. of 96% alcohol. The resulting precipitate is allowed to settle for about two hours, then filtered, suction-dried, washed four times with 25 ml. of alcohol, then twice with ether, and dried in air. Yield; 50-53 g. (75%).

Reprecipitation with alcohol is desirable. The pure preparation must give a perfectly clear solution in  $H_2O$ . To precipitate the salt, the alcohol is added from a wash bottle; during the addition, the flask is vigorously shaken to insure that the particle size of the precipitate will not be too small.

SYNONYM:

Sodium cobaltinitrite.

PROPERTIES:

Yellow crystalline powder. Very soluble in water, sparingly soluble in alcohol and ether. The aqueous solution is not stable and forms  $HNO_2$  and  $HNO_3$ . Crystal structure:  $J2_1$  type (cubic).

REFERENCE:

E. Bijlman. Z. analyt. Chem. 39, 284 (1900).

# Potassium Hexacyanocobaltate (III)

# K<sub>3</sub>[Co(CN)<sub>6</sub>]

The intermediate  $K_4[Co(CN)_6]$  is prepared from  $Co(CN)_2$  and KCN. Boiling of its solution precipitates  $K_3[Co(CN)_6]$ .

A clear, filtered solution of 48 g. of  $CoCl_2 \cdot 6H_2O$  in 500 ml. of  $H_2O$  is heated to boiling, and a clear solution of 30 g. of KCN in 200 ml. of H<sub>2</sub>O is added dropwise with vigorous stirring. Before adding the final portions of the KCN, a sample of the solution is filtered and the filtrate is treated with a drop of KCN solution, in order to establish whether any CoCl<sub>2</sub> is still present in the solution. The violet-red precipitate of Co(CN)<sub>2</sub> is filtered off, washed with cold H<sub>2</sub>O, and dissolved while still moist in a conc. solution of 60 g. of KCN. The deep red solution of  $K_4[Co(CN)_6]$ is heated to boiling for 10-15 min., whereupon it becomes yellow and evolves H<sub>2</sub>. If a small quantity of yellow  $K_3[Co(CN)_6]$  crystallizes at this time, some water is added to redissolve it. The boiling solution is filtered and cooled. The K<sub>3</sub>[Co(CN)<sub>6</sub>] precipitate is collected on a filter and washed with some cold water. Further quantities of the precipitate are obtained by concentrating the mother liquor to half its volume; this solid is worked up as above. The combined precipitates are recrystallized twice from hot water, some activated carbon being added to the solution. The pure, almost colorless crystals are filtered off and washed with some cold water. They must give a clear water solution.

The small excess of KCN called for in the directions for preparing the solution of  $Co(CN)_2$  in KCN prevents the precipitation of green  $K_2Co[Co(CN)_6]$ , which is insoluble at room temperature.

Alternate method: Oxidation of  $CoCl_2 \cdot 6H_2O + KCN$  in acetic acid solution with air, and several reprecipitations from acetic acid solution with alcohol [W. Biltz, W. Eschweiler and A. Bodensiek, Z. anorg. allg. Chem. <u>170</u>, 161 (1928)].

SYNONYM:

Potassium cobalt (III) hexacyanide.

PROPERTIES:

Formula weight 332.35. Small, almost colorless needles with a yellowish tinge.  $d_4^{25}$  1.878. Readily soluble in water; solubility in 87-88% alcohol (20°C): 1:7500 parts. Decomposes on heating in absence of air. Crystalline form: monoclinic, isomorphic with  $K_3$ Fe(CN)<sub>6</sub>.

REFERENCE:

A. Benedetti-Pichler. Z. anal. Chem. <u>70</u>, 257 (1927).

# Hexacyanocobaltic (III) Acid

H<sub>3</sub>Co(CN)<sub>6</sub>

 $\begin{array}{rrrr} K_{3}Co(CN)_{6} &+ & 3 \ HCl &= \ H_{3}Co(CN)_{6} &+ & 3 \ KCl \\ 332.4 & 109.4 & 218.1 & 223.7 \end{array}$ 

A solution of 3 g. of  $K_3Co(CN)_6$  in 9 ml. of  $H_2O$  is treated with 9 g. of conc. HCl. The KCl precipitate is removed by filtration.

The solution is sensitive to light.

SYNONYM:

Hydrogen hexacyanocobaltate (III).

REFERENCE:

A. von Baeyer and V. Villiger. Ber. dtsch. chem. Ges. <u>34</u>, 2687 (1901).

 $H_3Co(CN)_6 \cdot 5 H_2O$ 

A very small excess of  $H_2SO_4$  (d 1.84) is added to a 25% aqueous solution of  $K_3Co(CN)_6$ , which is then heated for 15-20 minutes to 50-55°C and cooled. Absolute alcohol is added and the alcohol-insoluble  $K_2SO_4$  is removed by filtration. The solution is carefully concentrated at 50-55°C and the  $H_3Co(CN)_6 \cdot 5 H_2O$  thus formed is recrystallized three times from alcohol.

PROPERTIES:

Colorless crystalline needles; hygroscopic. Heating at 100°C yields white  $H_3Co(CN)_6 \cdot 0.5 H_2O$ . At higher temperatures, colored products are formed until finally a black cobalt carbide remains. Not altered by brief boiling with HCl or HNO<sub>3</sub>, but forms Co<sub>3</sub>- $[Co(CN)_6]_2$  in hot  $H_2SO_4$  (d 1.84).

REFERENCE:

O. K. Dobrolyubskiy. Zh. Prikl. Khimii 26, 1185, 1233 (1953).

## Metallic Nickel

SABATIER METHOD

$$NiO + H_2 = Ni + H_2O$$
  
74.7 22.4 l. 58.7

Very pure,  $O_2$ -free  $H_2$  dried over  $P_2O_5$  is passed for 15 hours over NiO [obtained by thermal decomposition of Ni(NO<sub>3</sub>)<sub>2</sub> · 6 H<sub>2</sub>O] at 300-400°C. After cooling in the H<sub>2</sub> stream, the air-sensitive metal is transferred to small glass bulbs attached to the apparatus, and these are sealed off. The metal powder may also be stored in bottles under alcohol.

Used as hydrogenation catalyst.

PROPERTIES:

Atomic weight 58.71. Black metallic powder; pyrophoric. M.p. 1453°C, b.p. 3177°C. Ferromagnetic. Soluble in dil. HNO<sub>3</sub>, passivated by conc. HNO<sub>3</sub>.

**REFERENCE:** 

P. Sabatier. Die Katalyse in der organischen Chemie [Catalysis in Organic Chemistry], translated into German and enlarged by B. Finkelstein and H. Häuber, Leipzig, 1927.

# Nickel (II) Chloride

## NiCl<sub>2</sub>

I.

$NiCl_2$	$+ 6 SOCl_2$	$= \ NiCl_2$	+ 12 HCl	+ 6 SO <sub>2</sub>
(6 H₂O) 237,7	713.9	129.6	437.6	384 <b>.4</b>

Water is removed by refluxing with  $SOCl_2$  as described for  $CoCl_2$ .

п.

 $NiCl_2 \cdot 6 H_2O = NiCl_2 + 6 H_2O$ 237.7 129.6 108.1

The starting NiCl<sub>2</sub>  $\cdot$  6H<sub>2</sub>O is dried in a combustion tube at 150°C; the final heating to 400°C proceeds in a stream of Cl<sub>2</sub>-containing HCl. After the yellow NiCl<sub>2</sub> has formed, the tube is sealed at one end and the product is sublimed (oil-pump vacuum) at the highest temperature that the tube can withstand. To remove HCl, the NiCl<sub>2</sub> is annealed in high vacuum over KOH at 160°C.

Alternate methods: a) Heating of NiCl<sub>2</sub>  $\cdot$  6 H<sub>2</sub>O in a sealed tube with COCl<sub>2</sub> at 200°C (Hecht).

b) The frequently recommended thermal decomposition of NiCl<sub>2</sub>  $\cdot$  6 NH<sub>3</sub> does not yield pure NiCl<sub>2</sub>, since black by-products are formed.

c) Treatment of Ni(CH<sub>3</sub>COO)<sub>2</sub> ·  $4 H_2O$  with CH<sub>3</sub>COCl in benzene, analogous to the preparation of CoCl<sub>2</sub> [G. W. Watt, P. S. Gentile and E. P. Helvenston, J. Amer. Chem. Soc. 77, 2752 (1955)].

PROPERTIES:

Bright-yellow powder or crystalline leaflets (like mosaic gold). Subl. 993°C (760 mm.); m.p. 1001°C (in sealed tube);  $d_4^{25}$  3.521. Heat of formation: -73.0 kcal./mole (25°C). Sublimed NiCl<sub>2</sub> is relatively stable and takes up water slowly; fine NiCl<sub>2</sub> powder is hygroscopic and becomes green in air. Solubility in  $H_2O$  (g. NiCl<sub>2</sub>/100 g. solution): 34.8 (0°C); 40.4 (26.3°C); 46.7 (100°C). Moderately soluble in methyl and ethyl alcohol. Crystal structure: C 19 type.

**REFERENCES:** 

I. H. Hecht. Z. anorg. Chem. 254, 51 (1947).

II. W. Biltz and E. Birk. Z. anorg. allg. Chem. 127, 34 (1923).

## Hexaamminenickel (II) Chloride

## [Ni(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>2</sub>

 $NiCl_2 + 6 NH_3 = [Ni(NH_3)_6]Cl_2$ (6 H<sub>2</sub>O) 237.7 102.2 231.8

A conc. solution of cobalt-free NiCl<sub>2</sub>  $\cdot$  6H<sub>2</sub>O is treated with excess conc. NH<sub>3</sub>, then cooled in running water. The separation of fine crystals of [Ni(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>2</sub> is completed by addition of an ammoniacal NH<sub>4</sub>Cl solution. The precipitate is filtered off and successively washed with conc. ammonia, alcohol and ether.

Alternate methods: Action of  $NH_3$  on anhydrous NiCl<sub>2</sub> at room temperature [F. Ephraim, Z. phys. Chem. <u>81</u>, 513 (1913)].

PROPERTIES:

Blue-violet, fine crystalline powder.  $d_4^{25}$  1.468. Heat of formation: -16.3 kcal./mole. The decomposition pressure at 176.5 °C is 1 atm. Decomposes in H<sub>2</sub>O, liberating NH<sub>3</sub>. Soluble in aqueous ammonia; not soluble in conc. ammonia or alcohol. Crystal structure: J1<sub>1</sub> type.

**REFERENCE:** 

S. P. L. Sörensen. Z. anorg. Chem. 5, 354 (1894).

# Nickel (II) Bromide

#### NiBr<sub>2</sub>

ANHYDROUS NiBr<sub>2</sub>

 $Ni + Br_2 = NiBr_2$ 58.7 159.8 218.5

Nickel powder, produced by heating NiCl<sub>2</sub> in a hydrogen stream at 400°C, is covered with a layer of completely dry ether and then

treated with dry bromine. After 12 hours the ether is removed and the residue heated in vacuum at 130°C. In order to purify the preparation, which still contains some Ni, it is sublimed at 900°C (quartz or porcelain tube) in a  $CO_2$ -free stream of  $N_2$  + HBr.

Alternate methods: a) Heating of NiCl<sub>2</sub> in a stream of HBr at  $500^{\circ}$ C (G. Crut).

b) Heating of NiBr<sub>2</sub>  $\cdot$  6 H<sub>2</sub>O at 140°C in a drying oven [J. A. A. Ketelaar, Z. Kristallog. <u>88</u>, 26 (1934)].

c) Reaction of Ni(CH<sub>3</sub>COO)<sub>2</sub>  $\cdot$  4 H<sub>2</sub>O with CH<sub>3</sub>COBr in benzene, analogous to the preparation of CoCl<sub>2</sub> [G. W. Watt, P. S. Gentile and E. P. Helvenston, J. Amer. Chem. Soc. <u>77</u>, 2752 (1955)].

**PROPERTIES:** 

Yellow powder or bronze-yellow micalike particles. M.p. 963° (in sealed tube);  $d_4^{25}$  5.018. Heat of formation: -51.8 kcal. per mole (25°). Solubility in H<sub>2</sub>O (g. NiBr<sub>2</sub>/100 g. solution): 56.6 (19°C); 61.0 (100°C). Soluble in methyl and ethyl alcohols, acetone and quinoline; insoluble in toluene. Crystal structure: sublimed product, C19 type; unsublimed product, variable between C6 and V19 types.

**REFERENCE:** 

G. Crut. Bull. Soc. Chim. France [4] 35, 550 (1924).

NiBr<sub>2</sub> · 6 H<sub>2</sub>O

 $NiBr_2 + 6 H_2O = NiBr_2 \cdot 6 H_2O$ 218.5 108.1 326.6

A solution of  $NiBr_2$  in water is concentrated until crystallization begins and is then cooled. The crystals are filtered off and recrystallized from alcohol.

Alternate method: Precipitated nickel carbonate or  $Ni(OH)_2$  is dissolved in aqueous HBr, concentrated on the steam bath, and the product recrystallized from alcohol [J. A. A. Ketelaar, Z. Kristallog. <u>88</u>, 26 (1934)].

PROPERTIES:

Green crystals; transform to NiBr<sub>2</sub> · 3H<sub>2</sub>O at +28.5°C.

REFERENCE:

See G. L. Clark and H. K. Bruckner. J. Amer. Chem. Soc. <u>44</u>, 230 (1922).

# Nickel (II) Iodide

#### NiI<sub>2</sub>

ANHYDROUS Nil2

 $\begin{array}{rrrr} \text{Ni(OH)}_2 \ + \ 2 \ \text{HI} \ = \ \text{NiI}_2 \ + \ 2 \ \text{H}_2\text{O} \\ 92.7 & 255.8 & 312.5 & 36.0 \end{array}$ 

Either Ni(OH)<sub>2</sub> or precipitated nickel carbonate is dissolved in hydriodic acid and the solution evaporated to dryness. The solid is recrystallized from alcohol and dried at 140°C. A final sublimation in high vacuum at 500-600°C is recommended.

Alternate method: Dehydration of  $NiI_2 \cdot 6H_3O$  (Riedel, Thesis, Univ. of Halle, 1913).

PROPERTIES:

Black solid; forms small lustrous crystals when sublimed. M.p. 797°C (in sealed tube);  $d_4^{25}$  5.834. Heat of formation: -41.40 kcal./mole. Decomposes when heated to high temperatures in air. Hygroscopic; rapidly forms a green solution when exposed to air. Solubility in H<sub>2</sub>O (g. NiI<sub>2</sub>/100 g. solution): 57.8(11°C); 64.1 (43°C); 65.7 (90°C). Aqueous solutions of NiI<sub>2</sub> can dissolve up to two atoms of iodine, thereby acquiring a brown color. Dilute solutions are pure green, concentrated solutions dirty green or reddish brown. Slowly soluble in cold absolute alcohol, rapidly in hot. Crystal structure: C19 type.

REFERENCE:

J. A. A. Ketelaar. Z. Kristallogr. 88, 26 (1934).

Nil<sub>2</sub> · 6 H<sub>2</sub>O

 $NiI_2 + 6 H_2O = NiI_2 \cdot 6 H_2O$ 312.5 108.1 420.6

A solution of NiI<sub>2</sub> in  $H_2O$  is evaporated to a sirup. The crystals of NiI<sub>2</sub> •  $6 H_2O$  are filtered off and suction-dried.

Alternate method: As in the case of NiI<sub>2</sub>, but without the dehydration at 140 °C [J. A. A. Ketelaar, Z. Kristallogr. <u>88</u>, 26 (1934)].

PROPERTIES:

Blue-green crystals; rapidly deliquescent in air, becoming brown and giving off iodine. Exists up to 43°C; becomes anhydrous when heated on steam bath. **REFERENCES:** 

O. Erdmann. J. prakt. Chem [1] 7,254(1936); A. Étard. Ann. Chim. Phys. [7] 2, 503 (1894).

# Nickel (II) Oxide

#### NiO

Prepared by thermal decomposition of nickel salts of volatile acids.

I. Precipitated nickel carbonate or  $Ni(NO_3)_2 \cdot 6H_2O$  is heated in a Pt crucible for six hours at 1000-1100°C and allowed to cool in  $O_2$ -free nitrogen.

On cooling in air, a surface skin forms, which can be removed by postreduction with pure  $H_2$  at 100°C.

Test for active oxygen is essential.

II. Precipitated nickel carbonate is placed in a pear-shaped decomposition vessel, the air is displaced several times with  $O_2$ -free nitrogen, and the entire apparatus is degassed in high vacuum at 100°C. After high vacuum is established, the system is heated, over a period of 90 minutes, to 350°C. The product is transferred to previously prepared storage bulbs, which are then sealed.

The oxide blackens immediately in the presence of air. Test for active oxygen is essential.

Alternate method: Small chips of nickel are heated in air at 1000°C (for procedure, see method II, subsection on Cu<sub>2</sub>O) [H. H. von Baumbach and C. Wagner, Z. phys. Chem. (B) <u>24</u>, 61 (1934)]. Used as oxidation catalyst.

PROPERTIES:

Formula weight 74.71. M. p.  $1990^{\circ}$ C; d 6.67. Heat of formation: -58.4 kcal./mole. Bright-yellow powder, brown when heated. When the oxygen content is in slight excess, the color is dark olive green, becoming darker as the oxygen content increases. NiO prepared at high temperatures is almost insoluble in acids and alkalies; the lower the temperature of preparation, the more soluble it is, especially in hot nitric acid and ammonia. Crystal structure: B1 type.

**REFERENCE:** 

M. Le Blanc and H. Sachse. Z. Elektrochem. 32, 58(1926).

#### Nickel (II) Hydroxide

#### Ni(OH)2

 $Ni(NO_3)_2 + 2 NaOH = Ni(OH)_2 + 2 NaNO_3$ (6 H<sub>2</sub>O) 290.8 80.0 92.7 170.0

A solution of 25 g. of KOH in 250 ml. of carbonate-free  $H_2O$  is added dropwise and with vigorous stirring to a warm (about 35 °C) solution of 60 g. of Ni(NO<sub>3</sub>)<sub>2</sub> · 6 H<sub>2</sub>O in 250 ml. of H<sub>2</sub>O. The precipitate is washed by decantation with several five-liter portions of warm, CO<sub>2</sub>-free H<sub>2</sub>O (until the washings are no longer alkaline), then once with five liters of CO<sub>2</sub>-free H<sub>2</sub>O containing some ammonia, and finally with similar portions of warm, CO<sub>2</sub>-free water until both precipitate and washings are free of K<sup>+</sup> and NO<sub>3</sub><sup>-</sup>. The precipitate is filtered off and dried in a vacuum desiccator over conc. H<sub>2</sub>SO<sub>4</sub>.

The preparation still contains about one mole of adsorptively bound water, which can be removed by heating to 200°C.

Precipitation, decantation and filtration are carried out in the absence of  $CO_2$ . Another suitable starting material is Ni(NO<sub>3</sub>)<sub>2</sub> · 6 NH<sub>3</sub>; however, NiCl<sub>2</sub> · 6 H<sub>2</sub>O and NiSO<sub>4</sub> · 7 H<sub>2</sub>O are not recommended, since the precipitate then tenaciously retains Cl<sup>-</sup> and SO<sup>2</sup><sub>4</sub>.

#### PROPERTIES:

Apple-green crystalline powder; d 3.65. Decomposes at  $230^{\circ}$ C into NiO and H<sub>2</sub>O. Soluble in ammonia, ethylenediamine and acids. Crystal structure: C6 type. Heat of formation: -62.68 kcal./mole.

**REFERENCE:** 

G. F. Hüttig and A. Peter. Z. anorg. allg. Chem. 189, 183 (1930).

## β-Nickel (III) Hydroxide

#### NiO(OH)

A solution of 100 g. of Ni(NO<sub>3</sub>)<sub>2</sub>  $\cdot$  6 H<sub>2</sub>O in 1500 ml. of H<sub>2</sub>O is added dropwise and with vigorous stirring to a solution of 55 g. of KOH and 12 ml. of Br<sub>2</sub> in 300 ml. of H<sub>2</sub>O. The precipitation temperature should not exceed 25°C. The precipitate is washed five times (decantation) with  $CO_2$ -free  $H_2O$ , then several times (decantation) using a centrifuge, until  $NO_3^-$  and  $K^+$  can no longer be detected in either the precipitate or the wash water. The wet product is dried for three days over conc.  $H_2SO_4$ , then two weeks over 1:1  $H_2SO_4$ .

Precipitation, decantation and filtration must be carried out in a  $CO_2$ -free atmosphere.

Tests for active oxygen, Ni,  $H_2O$  and  $CO_2$  are necessary. The oxygen and  $H_2O$  contents vary. In view of the rapid conversion to Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub>, it is advisable to work up the precipitate as quickly as possible.

To determine active oxygen, a 100-mg. sample is covered with 100 ml. of  $H_2O$ , and 1 g. of KI and 25 ml. of 2 N  $H_2SO_4$  are added. The liberated  $I_2$  is titrated with Na<sub>2</sub>S<sub>2</sub>O<sub>3</sub>.

Alternate method: A solution of  $Ni(NO_3)_2 \cdot 6H_2O$  is treated with sodium acetate and electrolyzed at room temperature. The low yield is a disadvantage of this method [O. Glemser and J. Einerhand (1950)].

PROPERTIES:

Black powder.  $d_4^{20}$  4.15. Readily soluble in acids. Rapidly converted to Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub> by H<sub>2</sub>O and bases. Loses water when heated in vacuum and is converted to Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub>. Crystal structure: C 6 type.

REFERENCES:

G. F. Hüttig and A. Peter. Z. anorg. allg. Chem. <u>189</u>, 190 (1930);
 O. Glemser and J. Einerhand. Z. anorg. Chem. <u>261</u>, 26 (1950).

# γ-Nickel (III) Hydroxide

## NiO(OH)

Metallic Ni is fused with  $Na_2O_2$  + NaOH and the melt is extracted with  $H_2O_2$ .

A crucible of pure nickel is filled to one third of its volume with a mixture of one part of  $Na_2O_2$  and three parts of NaOH, and heated for four hours at 600°C. The melt is cooled, then carefully extracted with ice water, avoiding any rise in temperature. Washing by decantation with  $H_2O$  is carried out until there no longer is an alkaline reaction. The tiny crystals settle very easily; the flocculent precipitate is removed by slurrying.

Tests for active oxygen [see under  $\beta$ -NiO(OH)], Ni, and H<sub>2</sub>O are necessary.

Used as an oxidation catalyst.

PROPERTIES:

Small, lustrous, black hexagons or needles.  $d_4^{20}$  3.85. Soluble in dil.  $H_2SO_4$  with evolution of  $O_2$ . Decomposes on heating to 138-140°C. Crystal structure: resembles C 19 type.

**REFERENCE:** 

O. Glemser and J. Einerhand. Z. anorg. Chem. 261, 26 (1950).

#### Nickel (II,III) Hydroxide

Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub>

The precipitation of Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub> is carried out by dropwise addition of a solution of Ni(NO<sub>3</sub>)<sub>2</sub>  $\cdot$  6H<sub>2</sub>O to aqueous KOH + Br<sub>2</sub> at 50°C (as described under  $\beta$ -NiO(OH)). The product is washed by decantation with warm, CO<sub>2</sub>-free H<sub>2</sub>O. All operations are conducted in a CO<sub>2</sub>-free atmosphere.

Tests for active oxygen: see under  $\beta$ -NiO(OH). Analysis for Ni and H<sub>2</sub>O is recommended. Water and oxygen contents vary. Used as an oxidation catalyst.

Alternate method: Electrolysis  $(50-60^{\circ}C)$  of a solution of Ni(NO<sub>3</sub>)<sub>2</sub> treated with sodium acetate. The low yield is a disadvantage of this method (O. Glemser and J. Einerhand, see reference below).

PROPERTIES:

Formula weight 276.16. Black powder.  $d_4^{20}$  3.33. Readily soluble in acids. On heating to 140°C, converts to NiO, H<sub>2</sub>O and O<sub>2</sub>. Crystalline form: hexagonal.

**REFERENCES:** 

O. Glemser and J. Einerhand. Z. anorg. Chem. 261, 26 (1950).

#### Nickel (II) Sulfide

NiS

 $\alpha$ -NiS

 $NiCl_2 + H_2S = NiS + 2 HCl$ (6 H<sub>2</sub>O) 237.7 22.1 l. 90.8 72.9

The apparatus used for precipitation of nickel sulfides in the absence of air is shown in Fig. 335. The  $\alpha$ -NiS is prepared



Fig. 335. Preparation of nickel sulfides in the absence of air.

as follows. The air is displaced from the apparatus by a stream of  $CO_2$ -free,  $O_2$ -free nitrogen. At the same time, all liquids in the apparatus are boiled and allowed to cool in the  $N_2$  stream. These liquids comprise the solution in a, which is 0.4 N in NiCl<sub>2</sub> and 0.8 N in NH<sub>4</sub>Cl; the water in b, which covers the chunks of FeS; the water in wash bottle c; the saturated NH<sub>4</sub>Cl, with excess of solid NH<sub>4</sub>Cl, in d; the conc. KOH in the bubble-counting tube e; and the wash water in f. The mercury trap h serves to prevent the entrance of air into the reactor. Now CO<sub>2</sub>-free H<sub>2</sub>S is generated in b, and the sulfide precipitates in a. The wash water is transferred into a (N<sub>2</sub> pressure), shaken with the precipitate, and (after the latter has settled) drawn off through fritted disc g, using a vacuum pump. The washing procedure is repeated 15-20 times.

The NiS should not be completely precipitated from the solution, otherwise it does not settle well during washing.

PROPERTIES:

Black powder, soluble in HCl. Converts to Ni(OH)S in air. Amorphous on x-ray analysis. Heat of formation: -19.37 kcal./mole.

**REFERENCE:** 

E. Dönges. Z. anorg. Chem. 253, 345 (1947).

 $\beta$ -NiS

Ni + S = NiS58.7 32.1 90.8

A stoichiometric mixture of Ni and S is heated for six hours at 900°C in a sealed, evacuated quartz tube.

Alternate methods: a) Precipitation with  $H_2S$  from a 1 N NiCl<sub>2</sub> solution containing acetic acid, workup as for  $\alpha$ -NiS [A. Thiel and H. Gessner, Z. anorg. Chem. <u>86</u>, 1 (1914)].

b) The alpha form of NiS is digested with 0.2 N acetic acid for several days in the absence of air (A. Thiel and H. Gessner, see above).

PROPERTIES:

Black powder. M.p. 810°; d 5.0-5.6. Dissolves rapidly on boiling in 2 N HCl. Crystal structure: B8 type.

**REFERENCE:** 

W. Klemm and W. Schüth, Z. anorg. allg. Chem. 210, 39 (1933).

γ-NiS

 $NiSO_4 + H_2S = NiS + H_2SO_4$ (7 H<sub>2</sub>O)
280.9 22.1 l. 90.8 98.1

The apparatus shown for  $\alpha$ -NiS is used and H<sub>2</sub>S is bubbled through 1 N NiSO<sub>4</sub> solution weakly acidified with dil. H<sub>2</sub>SO<sub>4</sub>. Air must be rigorously excluded during the reaction. The precipitate is worked up in the same way as  $\alpha$ -NiS.

PROPERTIES:

Black powder.  $d_4^{30}$  5.34. Converts to  $\beta$ -NiS at 396°C. Crystal structure: B13 type.

REFERENCE:

R. G. Levi and A. Baroni. Z. Kristallogr. 92, 210 (1935).

DRYING OF PRECIPITATED NiS

The slurry of the sulfide is dried in a drying pistol for eight hours (aspirator vacuum); the temperature is slowly raised to 150°C. Then NaOH is placed in the drying pistol and the material is dehydrated for four additional hours at 180°C in an oil-pump vacuum. The sulfide is then transferred to another small flask and dried for 5-12 hours at 300-400°C (high vacuum). The latter operation must be carried out with care, to avoid dusting of the product.

**REFERENCES:** 

W. Biltz. Z. anorg. allg. Chem. 228, 275 (1936); 239, 82, 126 (1938).

## Nickel (IV) Sulfide

#### NiS<sub>2</sub>

 $NiS + S = NiS_2$ 90.8 32.1 122.8

Completely dry NiS is heated in a sealed tube with triple distilled S (five hours at 450°C). Four to five times the stoichiometric amount of S is used. After the reaction is complete, the excess S is removed by extraction with  $CS_2$  in a Soxhlet apparatus.

Test for S: The sulfide is dissolved in HNO<sub>3</sub>, and  $BaCl_2$  is added to cause precipitation.

PROPERTIES:

Black to gray powder. d 4.39. Sensitive to air even when dry, evolving  $SO_2$ , which remains partly adsorbed. Can incorporate S into the crystal lattice. Soluble in nitric acid. Crystal structure: C 2 type.

**REFERENCES:** 

W. F. deJong and H. W. V. Willems. Z. anorg. allg. Chem. <u>160</u>, 185 (1927); W. Biltz. Ibid. 228, 278 (1936).

# Nickel (II) Amide

#### Ni(NH<sub>2</sub>)<sub>2</sub>

 $\frac{\text{Ni}(\text{SCN})_2}{174.9} + \frac{2 \text{ KNH}_2}{110.2} + \frac{\text{Ni}(\text{NH}_2)_2}{90.8} + \frac{2 \text{ KSCN}}{194.4}$ 

The apparatus shown in Fig. 334 [see subsection on  $Co(NH_2)_3$ ] is charged with an excess of dry Ni(SCN)<sub>2</sub>. Then a solution of KNH<sub>2</sub> in liquid NH<sub>3</sub> is added. A flocculent red precipitate forms; this is washed with liquid NH<sub>3</sub> until peptization just starts. The product is transferred to an auxiliary apparatus (as in the procedure on p. 1527) and dried in high vacuum at 40°C.

Used for the preparation of  $Ni_3N_2$ .

1554

PROPERTIES:

Red powder. Reacts mildly with  $H_2O$  to form Ni(OH)<sub>2</sub> and NH<sub>3</sub>. Decomposes when heated in vacuum to 120°C.

**REFERENCE:** 

G. S. Bohart. J. Phys. Chem. 19, 537 (1915).

## Trinickel Dinitride

#### $Ni_3N_2$

 $3 \operatorname{Ni}(\operatorname{NH}_2)_2 = \operatorname{Ni}_3\operatorname{N}_2 + 4 \operatorname{NH}_3$ 272.3 204.1 88.3 l.

Nickel (II) amide (see preceding preparation) is heated in vacuum at 120°C;  $NH_3$  is slowly evolved and  $Ni_3N_2$  is formed. Some  $N_2$  is also produced in a side reaction.

Alternate method: A mixture of 10 parts of NiO and 7.4 parts of completely anhydrous Ni(CN)<sub>2</sub> is fused in an electric arc surrounded by pure N<sub>2</sub> [A. C. Vournasos. Comptes Rendus Hebd. Séances Acad. Sci. <u>163</u>, 889 (1919)].

PROPERTIES:

Black powder. Reacts very slowly with water. Decomposes to Ni and N<sub>2</sub> above 120°C. Alkalies liberate  $NH_3$ .

**REFERENCE:** 

G. S. Bohart. J. Phys. Chem. 19, 537 (1915).

# **Trinickel Nitride**

#### Ni<sub>3</sub>N

An alumina boat is charged with 20 mg. of nickel obtained from Ni(CO)<sub>4</sub>. The boat is heated for three hours at 445°C in a reaction tube through which flows a stream of  $NH_3$  (22 cm./sec.). The preparation is cooled in the  $NH_3$  stream, ground carefully in an agate mortar, and allowed to react with  $NH_3$  once more under the same conditions.

The test for N in the product may be made by the micro-Kjeldahl method.

PROPERTIES:

Black-gray powder.  $d_4^{25}$  7.66. Unaffected by moisture and O<sub>2</sub>. Dissolved slowly by dil. mineral acids in the cold, rapidly by conc. HCl and conc. HNO<sub>3</sub>. Dissolved rapidly by all hot acids. Not attacked by aqueous NaOH. Crystal structure: hexagonal close packing of Ni atoms, oriented incorporation of N.

REFERENCE:

R. Juza and W. Sachse. Z. anorg. allg. Chem. 251, 201 (1943).

# Nickel Carbide

## Ni<sub>3</sub>C

 $3 \text{ Ni} + 2 \text{ CO} = \text{ Ni}_{8}\text{C} + \text{ CO}_{2}$ 176.1 44.8 l. 188.1 22.3 l.

Pure NiO is reduced with pure  $H_2$  at 275-285°C until constant weight. The fine Ni powder is heated at once with pure CO (completely free of O<sub>2</sub>) for 260 hours at 270°C. The Ni<sub>3</sub>C thus produced is pyrophoric. This may be remedied by heating for a long time in O<sub>2</sub>-free nitrogen at 250°C and cooling in the N<sub>2</sub> stream. Test for bound C: Heating with pure H<sub>2</sub> at 250-270°C evolves

Test for bound C: Heating with pure  $H_2$  at 250-270°C evolves C as  $CH_4$ .

PROPERTIES:

Gray-black powder.  $d_4^{18}$  7.97. Heat of formation: -9.2 kcal. per mole. Decomposed at room temperature by conc. and dil. HCl; precipitation of C does not occur (see Fe<sub>3</sub>C). Soluble in dil. HNO<sub>3</sub>; dil. H<sub>2</sub>SO<sub>4</sub> causes separation of C. Stable at temperatures up to 380-400°C. Crystal structure: hexagonal close packing of Ni atoms.

**REFERENCE:** 

H. A. Bahr and Th. Bahr. Ber. dtsch. chem. Ges. 61, 2177 (1928).

# Nickel (II) Carbonate

 $NiCO_3 \cdot 6 \ H_2O$ 

Prepared by electrolysis of  $CO_2$ -saturated  $H_2O$  with nickel electrodes.

An electrolysis cell  $(20 \times 15 \times 20 \text{ cm.})$  is covered with a wooden lid from which three pieces of nickel sheet electrodes  $(20 \times 12.7 \times 0.05 \text{ cm.})$  are suspended. Two of the electrodes are placed at the sides of the vessel and are interconnected; the third is in the center and serves as the anode. The cell is filled with conductivity water to 2.5 cm. below the top, pure CO<sub>2</sub> is bubbled through, and the current is turned on and controlled at 2-2.2 amp. The cell is cooled externally with running water. The NiCO<sub>3</sub> · 6 H<sub>2</sub>O drops to the bottom. It is filtered off and dried at 100°C. The yield is about 30 g./day.

PROPERTIES:

Formula weight 226.82. Pale-green rhombohedra or monoclinic prisms. Readily soluble in acids. The product prepared in the above manner is free of alkali, but contains some black hydroxide.

**REFERENCE:** 

E. C. C. Baly and N. R. Hood. Proc. Roy. Soc. London <u>122</u>, 313 (1929).

NiCO<sub>8</sub>

Anhydrous  $NiCO_3$  exists in two forms, one green (I) and one yellow (II).

GREEN FORM (I):

A solution of 0.12 moles of NiCl<sub>2</sub> in 100 ml. of water is acidified with HCl and charged into an autoclave. At 250°C and a CO<sub>2</sub> pressure of 1700 p.s.i., a solution of 0.18 moles of NaHCO<sub>3</sub> in 100 ml. of water is added dropwise. Green, crystalline NiCO<sub>3</sub> precipitates. Yield: 25-30%.

YELLOW FORM (II) :

The yellow NiCO<sub>3</sub> forms under the same conditions as the green, but at a temperature of  $180^{\circ}$ C and from very conc. solutions (0.22 moles of NiCl<sub>3</sub> in 25 ml. of water and 0.38 moles of NaHCO<sub>3</sub> in 100 ml. of water). The yield is poor. At lower temperatures only colloidal products are obtained; at temperatures between  $180^{\circ}$  and  $250^{\circ}$ C a mixture of I and II is produced.

PROPERTIES:

The green form consists of microscopically small, transparent, doubly refracting green rhombohedra; it is not attacked by warm conc. acids or by boiling water. The habit and physical and chemical properties of II are the same as those of I. On heating to approximately 400°C, both carbonates decompose to  $CO_2$  and green NiO. Both crystallize as rhombohedra.

**REFERENCE:** 

R. de St. Léon Langlés. Ann. Chimie 7, 568 (1953).

# Nickel (II) Thiocyanate

# Ni(SCN)2

Prepared by dissolving Ni(OH)<sub>2</sub> in a dilute HNCS solution and evaporating the resulting solution of Ni(SCN)<sub>2</sub>.

A dilute HNCS solution is saturated with Ni(OH)<sub>2</sub> or nickel carbonate and the deep-green solution is evaporated at about 15°C. Large green crystals of Ni(SCN)<sub>2</sub> · 4H<sub>2</sub>O are deposited. Above 15°C and on drying of Ni(SCN)<sub>2</sub> · 4H<sub>2</sub>O, a yellow powder of Ni(SCN)<sub>2</sub> · 0.5 H<sub>2</sub>O is obtained. It can be rendered anhydrous by heating to 150°C.

PROPERTIES:

Formula weight 174.88. Dark chocolate-colored powder. On addition of water, becomes first yellow and then dissolves with a green color.

**REFERENCES:** 

A. Rosenheim and R. Cohn. Z. anorg. Chem. <u>27</u>, 280 (1901); Ber. dtsch. chem. Ges. <u>33</u>, 1111 (1901); A. de Sweemer. Natuurwetensch. Tijdschr. <u>14</u>, 231 (1932).

# Di-µ-sulfido-tetrakis(dithiobenzoato)dinickel (IV)

$$(C_6H_5 \cdot CSS)_2NiS_2Ni(SSC \cdot C_6H_5)_2$$

This nickel (IV) complex compound is obtained by oxidation of the corresponding nickel (II) complex compound with  $O_2$ .

An alcoholic solution of 1 mmole (0.13 g.) of NiCl<sub>2</sub> is treated with an alcoholic solution of 12 mmoles (0.15 g.) of monothiobenzoic acid and 6 mmoles (0.24 g.) of NaOH. The clear yellowbrown solution (in which the Ni is present as  $[Ni(SCOC_6H_5)_4]^{2-}$ ) is refluxed at 50°C, and at the same time a gentle stream of  $O_2$ is introduced. The oxidation is shown by a change in color, first to red and then to violet. The reaction is complete in about four hours. The dark-violet compound is filtered off, washed with alcohol and water, and recrystallized from benzene.

PROPERTIES:

Formula weight 794.55. Tuft-shaped aggregates of dark-violet crystals. May be recrystallized from benzene, alcohol, ether and  $CS_2$ . Very stable to acids and bases; decomposed only by oxidizing acids.

REFERENCE:

W. Hieber and R. Brück. Z. anorg. allg. Chem. 269, 26 (1952).

Potassium Tetracyanonickelate (II)

## $K_2[Ni(CN)_4] \cdot H_2O$

NiS	O4	+ 2 KCN	= Ni(C)	$N)_2 + K_2SO_4$	
(6 H 262	$^{2}O)$	130.2	110.	7 174.3	
Ni <b>(</b> CN)2	+	2 KCN +	$H_2O =$	K <sub>2</sub> [Ni(CN) <sub>4</sub> ] ·	H <sub>2</sub> O
110.7		130.2	18.0	259.0	

A solution of 60 g. of NiSO<sub>4</sub>  $\cdot$  6 H<sub>2</sub>O in 200 ml. of water is prepared, and a solution of 29.7 g. of KCN in 70 ml. of water is added slowly, with constant stirring. The gray-green precipitate of Ni(CN)<sub>2</sub> is washed until free of sulfate and then filtered off.

The solid Ni(CN)<sub>2</sub> is placed in a solution of 29.2 g. of KCN in about 30 ml. of water. The solution, which is now red, is heated on a hot plate until small crystals appear. These are redissolved and the solution is allowed to cool. The compound precipitates as beautiful crystals. The yield is 57.4 g. (97%).

SYNONYM:

Potassium nickel (II) cyanide hydrate.

PROPERTIES:

Orange-red crystals. The water of hydration is completely removed by heating to  $100^{\circ}$ C. Very soluble, even in cold water; decomposed to Ni(CN)<sub>2</sub> by mineral acids. Forms black precipitates of higher nickel hydroxides on addition of hypobromites.

REFERENCE:

W. C. Fernelius and J. J. Burbage in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 227.

# SECTION 29

The Platinum Metals

H. L. GRUBE

# Pure Platinum

#### Pt

Platinum obtained from Russian or Colombian platinum ores or by industrial recovery processes contains the other platinum metals as impurities, as well as gold, iron and copper.

The following methods are recommended for its purification: the lead fusion process, based on the analytical procedures of Saint-Claire-Deville and Stas, and the caustic soda precipitation of Schneider [1] and Seubert [2], in which all the platinum metals except Pt pass into their lower oxidation states, which are not precipitated by  $NH_4Cl$ . There is also the Finkener process, based on recrystallization of  $Na_2PtCl_6$  from dilute soda solutions; this has been described by Mylius and Förster [3]. However, this method leads to poor yields because of the low solubility of  $Na_2PtCl_6$ . A process by Reerink [4] makes use of platinum's ability to form a volatile carbonyl chloride with  $Cl_2$  and CO.

The hypochlorite method developed by Mylius and Mazzucchelli [5] for the preparative purification of the platinum metals can be recommended as a laboratory procedure. Platinum (in the form of small foil clippings or sponge) is dissolved in the purest aqua regia available (glass or porcelain vessels); the solution is withdrawn (or decanted) and carefully evaporated in a porcelain dish over a small flame; the concentrate is dissolved in hydrochloric acid and hot  $H_2O$ . The chloride solution is diluted with a large amount of water, heated to about 80°C, and made slightly alkaline with soda. Any IrO<sub>2</sub> present is precipitated with Cl<sub>2</sub>, bubbled through for a short time. The initially colloidal iridium-containing precipitate coagulates after a short time to a black flocculent deposit which settles rapidly in the yellowish-red solution. Addition of a few milliliters of alcohol produces a marked increase in the speed of separation. The more carefully the solution is neutralized (without, however, going below a pH of 7), the more complete is the separation of the unwanted oxide.

The other platinum metals, gold and the heavy base metals may be precipitated as the oxides by addition of hypochlorite. This reaction can therefore be used for the removal of all metallic impurities. The only oxide soluble in an excess of hypochlorite is the black  $RuO_2$  (it is thereby converted to the volatile  $RuO_4$ ).

The filtered platinum solution is heated in a beaker or porcelain vessel and treated with  $NH_4Cl$ . The  $(NH_4)_2PtCl_6$  deposit is then filtered off and extracted by boiling with distilled water to dissolve any small quantities of  $(NH_4)_2PdCl_6$  still remaining.

The very pure platinum obtained after ignition is free of all the other platinum metals, gold and the heavy base metals. If the Pt still contains a few tenths or hundredths of a percent of Ir, the purification process can be repeated.

REFERENCES:

- 1. W. von Schneider. Liebigs Ann. 5, 271 (1997).
- 2. K. Seubert. Ibid. 207, 8 (1881).
- 3. F. Mylius and F. Förster. Ber. dtsch. chem. Ges. 25, 665 (1892).
- 4. E. H. Reerink. Z. anorg. allg. Chem. 173, 45 (1928).
- 5. F. Mylius and A. Mazzucchelli. Ibid. 89, 1 (1914).

## **Reclaimed Platinum**

In order to reclaim platinum residues from the laboratory (e.g., platinum absorbed on filter papers, scraps, filter ash, etc.), these are well ignited and then sieved through a fine screen, separating the residues into "fine" and "coarse" fractions.

The coarse fraction remaining on the screen contains the metallic residues, such as small pieces of wire, small clippings and pieces of foil. It is advisable to pass a magnet through this material to locate and remove any iron present in the form of nails, wire, etc. In some cases, it may be advisable to extract the iron from this fraction by boiling with dilute hydrochloric acid. Copper or brass residues are extracted by boiling with dilute nitric acid. The coarse material is then dissolved in aqua regia and processed to recover the Pt and its associated metals. In order to expel the nitric oxides present, the aqua regia solution is evaporated to a sirupy thickness, taken up with water and some dil. hydrochloric acid, and treated with  $NH_4Cl$  to precipitate the  $(NH_4)_2PtCl_6$ . The Pt still remaining in the mother liquor is precipitated with pure Zn.

The iron in the fine fraction is first extracted by boiling with dilute hydrochloric acid; after ignition, the residue is dissolved by heating in aqua regia. This solution is evaporated to a sirupy thickness and taken up with  $H_2O$  and HCl; the Pt is precipitated as  $(NH_4)_2PtCl_6$ .

If aged solutions or more or less reduced salts and preparations are to be processed, Berthold's work-up method is recommended. Often dirt appears in the residue on long standing, and sometimes the Pt salts are partially reduced by alcohol on long storage. In either case the liquid is first filtered, the residue is treated with aqua regia to dissolve any platinum it may contain, and the resulting solution is filtered.

The last filtrate is not combined with the first one, but is evaporated to drive off the aqua regia. The residue is extracted with hot water and this solution added to the main (that is, the first) filtrate.

When the solution to be worked up originates from an analytical laboratory, the main impurities are likely to be salts of K, Na, Mg and NH<sub>4</sub>. Alcohol and ether may also be present. In this case, the liquid is treated with some conc. HCl and pure Zn. If  $K_2PtCl_6$  precipitates, it must be reduced by heating to convert it into soluble material. Any alcohol or ether present must first be driven off.

After the reduction, which is clearly indicated by the decoloration of the liquid, the supernatant is decanted; the residue is thoroughly extracted by boiling with conc. HCl and washed by decantation with hot distilled water until the wash water no longer contains any chloride.

Double salts containing a platinum and an alkali metal ion (and especially the ammonium ion) are best treated by careful calcination in a Pt crucible under a layer of  $NH_4Cl$ , extraction with boiling water acidifed with some hydrochloric acid, and reignition.

For reclaiming procedure for Pt from electrolytic Pt baths, see p. 1567.

REFERENCE:

A. Berthold. Z. angew. Chem. <u>14</u>, 621 (1901).

# Platinum Sponge

Platinum sponge is best produced by prolonged ignition of  $(NH_4)_2PtCl_6$  in a Pt dish or bowl (dull red heat, about 600°C). The ignited sponge should be boiled with dilute hydrochloric acid, and then with distilled H<sub>2</sub>O. Finally it is gently heated again in a Pt dish.

# **Platinum Black**

Of the various methods of preparation given in the literature, that described by Gutbier and Maisch seems to be the best. A 5%

solution of  $H_2PtCl_6$  is heated, neutralized with  $Na_2CO_3$ , and poured into a boiling solution of sodium formate. The black residue which precipitates immediately is washed by decantation with hot  $H_2O_2$ . It is then filtered off with suction and freed from residual water by pressing between filter papers; it is further dried over  $P_2O_5$  or conc.  $H_2SO_4$ .

PROPERTIES:

Black powder, very active toward  $H_2$ , with a maximum absorptive capacity for  $H_2$  at 0°C.

REFERENCE:

A. Gutbier and O. Maisch. Ber. dtsch. chem. Ges. 52, 1370 (1919).

# Platinized Asbestos

Asbestos is saturated with an alcoholic solution of  $H_2PtCl_6$ . 6  $H_2O$  (technical "platinum chloride"), thereby producing a material with a definite platinum concentration. In view of the cost of the solution, the calculated quantity of  $H_2PtCl_6$  must be absorbed quantitatively by the asbestos; it is therefore essential to establish accurately (by preliminary experiments) the absorbance of the asbestos to be used. The  $H_2PtCl_6$ -saturated asbestos is kneaded as uniformly as possible and the mass is ignited while being agitated with a Pt spatula or rod. This method is particularly suitable for producing asbestos with low Pt contents (0.1-1% Pt). Its advantage lies in that the product contains no foreign salts which could obstruct the pores of the asbestos fibers and adversely affect its catalytic activity.

To prepare platinized asbestos with high platinum contents (5-10%), the mass is saturated with a H<sub>2</sub>PtCl<sub>8</sub> solution which does not contain alcohol. The procedure is the same as described above. The mass is made slightly alkaline by treating it with dil. sodium hydroxide, and the chloride is then reduced to fine, particulate Pt with sodium formate. The reduction is best carried out in a muffle furnace at about 300-400°C. Finally the asbestos is freed of alkali salts by thorough washing with cold water and is dried in a muffle furnace. At this point the asbestos should be light gray.

Erdmann gives another method. Asbestos saturated with conc.  $H_2PtCl_6$  is placed in a conc. solution of  $NH_4Cl$ . The asbestos, which is thus permeated with  $(NH_4)_2PtCl_6$ , is placed on a glass funnel to allow the excess solution to drain, and is then slowly heated to incandescence. This produces an asbestos with a high concentration of platinum sponge; however, the uniformity of the product leaves much to be desired.

For analytical purposes the platinized asbestos fibers should be as short as possible (almost powdery). For large-scale catalytic processes, the fibers should, on the other hand, be as long as possible.

REFERENCE:

O. Erdmann. Lehrb. d. anorg. Chem. [Inorganic Chemistry Text], 5th ed., p. 175 (1910).

# Handling of Platinum Equipment

Because the various chemicals used in ignition and melting processes may be corrosive, certain precautions must be observed in using platinum apparatus. Unfortunately, the use of platinum as vessel material is not a panacea for all the corrosion problems that plague the chemist.

Materials which readily form alloys with Pt (nonmetals P, As, Te, Si, B and C) or metals which melt at low temperatures (Pb or Sn) or substances which liberate these materials during ignition or melting processes can not only damage but even destroy platinum apparatus. This also holds for all melts containing potassium hydroxide, sodium nitrate, or mixtures of the alkali hydroxides or alkali carbonates with sodium nitrate; melts containing peroxides, cyanides or sulfides are particularly injurious to crucibles.

In general, ignitions should not be carried out at unnecessarily high temperatures or with reducing flames; reduction with an acetylene flame is forbidden.

Reducing conditions involving burner or flue gases, activated charcoal and the like are particularly deleterious when free silicic acid is also present. In this case, platinum-silicon alloys are formed, leading to the characteristic silicon fracture.

The critical corrosion temperature, i.e., the temperature above which serious corrosion occurs, generally lies around 700°C. However, this temperature is about 500-600°C for melts consisting mainly of KOH, Ba(OH)<sub>2</sub>, peroxides or cyanides; for melts composed mainly of carbonates or neutral salts, it is 800°C or somewhat higher. Further details are given in the Mitteilungen aus dem Chem. Laboratorium, W. C. Heraeus Co., Hanau; publications of the Engelhardt Industries, Newark, N. J.; G. Bauer, Chemiker-Ztg. 62, 257 (1938).

## CLEANING

Careful treatment of Pt vessels after use is essential. Usually the crucible contents can be easily removed by mechanical means. In some cases, the contents can be dissolved with warm hydrochloric acid or chlorine-free nitric acid. If this does not suffice, sodium pyrosulfate is heated in the Pt crucible until liberation of  $SO_3$ , the molten liquid is poured out, and the material still adhering to the walls is dissolved with hot water (O. Brunck). The crucible is not damaged at all by this treatment. If the edges of the crucible lid become bent out of shape, they can be smoothed out against a glass plate with a spatula made of horn or plastic. A very badly dented crucible should best be repaired by a goldsmith or other expert.

A Pt crucible that develops a small tear due to careless handling can readily be repaired in the laboratory. A small piece of thin gold foil of suitable size and shape is placed over the tear, and the spot is heated with an oxyhydrogen torch until the gold melts, after which the patch is smoothed with a burnisher. Such a crucible can still be used for most purposes. Very small tears can be healed with an oxyhydrogen torch; there is no need for a patch material in this case. Platinum wires, e.g., the leads to thermocouples or electrodes, can also be welded together quite simply with an oxyhydrogen torch: the two wires to be joined are laid close together and fused, or are welded together at a slightly lower temperature by a light tap with a hammer.

In case of serious damage, however, it is advisable to have the vessel repaired by a specialist.

# **Platinum Electroplating**

Thin layers of platinum can easily be deposited from electrolytic baths; however, the deposition must be repeated several times to produce thicker deposits.

Böttger gives a very good bath formula: it consists of a solution of  $(NH_4)_2PtCl_6$  in sodium citrate. Langbein gives the following instructions for preparing this bath: 500 g. of citric acid is dissolved in two liters of  $H_2O$  and neutralized with sodium hydroxide. This solution is heated to boiling and the  $(NH_4)_2PtCl_6$ , freshly precipitated from a solution of 75 g. of dry  $H_2PtCl_6$ , is added with stirring; the mixture is heated until the  $(NH_4)_2PtCl_6$  is completely dissolved; the solution is then cooled and diluted with  $H_2O$  to five liters. To reduce the electrical resistance of the bath, 4-5 g. of  $NH_4Cl$  is added per liter.

The electrolysis proceeds at 3-4 volts, a current density of 0.065 amp./in.<sup>2</sup> and a temperature of 70-90 °C. Thin Pt sheet is always used as the anode in platinum plating baths; it is scarcely attacked at all. The bath must always be kept slightly alkaline; if this is no longer the case after prolonged passage of current, dilute aqueous ammonia is added until an odor of ammonia is

noticeable. If the bath is acid, Pt sponge will rapidly precipitate.

C. W. Keitel et al. give another bath formula: cis-dinitrodiammineplatinum is dissolved in ammonia, and ammonium nitrate and sodium nitrite are added to improve the conductivity of the bath.

Objects to be plated with platinum (cathodes) are prepared in exactly the same way as in any other method of electroplating. Given the cost of platinum, these objects are usually small; platinum electrolysis vessels therefore need to hold only a few liters of the solution; the vessel will thus usually be a glass or porcelain beaker or a small iron trough coated on the inside with a special alkali- and acid-resistant enamel. If the platinum deposit is too dull, it can be rubbed and scoured in the same way as gilt-ware, and then replaced in the platinum bath to deposit a further platinum layer; this treatment may be repeated until the required deposit thickness is reached.

The current efficiency in platinum coating is very small (only 30-40% of the theoretical) since large quantities of energy are consumed in liberating the great amount of H<sub>2</sub> that evolves. This H<sub>2</sub> also hardens the platinum deposits.

**REFERENCE:** 

Pfanhauser et al. Galvanotechnik [Applied Electrochemistry], Leipzig, 1949, pp. 939-942; gives further details on bath compositions and on other recently developed plating baths.

According to Lummer and Kurlbaum, the liquid most suitable for the platinization of electrodes for potential measurements consists of 3 g. of  $H_2PtCl_6 \cdot 6 H_2O$  plus 0.10 g. of lead acetate in 97 ml. of  $H_2O$ ; the bath temperature is 20-30 °C, the potential about 4 volts. Two series-connected storage batteries are used as a power source and the current is regulated to produce a moderate gas evolution. The current is reversed, so that each electrode serves alternately as the anode and cathode. The total time for initial platinization is 10-15 minutes; usually only 1-2 minutes is sufficient for replating electrodes already covered with platinum black (before platinum plating these electrodes must be carefully cleaned; it is best to do this with chromosulfuric acid).

A thin coat of platinum (burnished platinum) can be produced on glass and porcelain by baking on either of the following special solutions:

I. A solution of 1 g. of platinic chloride  $(H_2PtCl_6 \cdot 6 H_2O)$  in 3.5 ml. of absolute alcohol is mixed with 10 ml. of a concentrated alcoholic solution of boric acid and 25 ml. of a solution of Venetian turpentine in lavender oil.

II. A solution of 1 g. of platinic chloride  $(H_2PtCl_6 \cdot 6 H_2O)$  in the minimum amount of absolute alcohol is added slowly (stirring) to

6 ml. of ice-cold lavender oil. After warming, Burgundy pitch is added to the mixture to give the required consistency.

In either case, the platinum-containing mass is spread uniformly on the glass (or porcelain) and carefully heat-dried so that no bubbles develop. The coated surface is then heated to a dull red heat in a muffle furnace or in a sulfur-free blowtorch flame.

The ingredients of these solutions are not usually available in the laboratory. However, ready-for-use solutions for producing burnished platinum (with instructions) are provided by companies handling noble metals (e.g., W. C. Heraeus of Hanau or Degussa of Frankfurt or Engelhardt Industries of Newark, N. J.).

**REFERENCE:** 

Ostwald-Luther. Physiko-Chemische Messungen [Physicochemical Measurements], 4th ed., Leipzig, 1925, pp. 158-159.

## RECLAIMING PLATINUM FROM USED BATHS

If the liquid quantity is not too large, then the best method is to precipitate the platinum with  $H_2S$  (this is preferred over the procedure involving concentration of the solution and reduction of the residue to the metal). The platinum can be precipitated from larger liquid quantities with pure Zn (following acidification of the bath liquid).

# Platinum Chlorides

Streicher and Krustinsons gives the following facts about the stability of chlorides of platinum as a function of temperature at 1 atm. of  $Cl_2$  pressure:

PtCl₄ (russet) stable up to 382°C,

PtCl<sub>3</sub> (dark green) stable between 382 and 435 °C,

PtCl<sub>2</sub> (greenish brown) stable between 435 and 515°C.

The existence of PtCl (pale yellow-green), 581-583°C, is not altogether certain.

**REFERENCES:** 

# S. Streicher. Thesis, Univ. of Darmstadt, 1913; J. Krustinsons. Z. Elektrochem. <u>44</u>, 537 (1938).

**PtCl₄** 

Prepared by decomposition of  $H_2PtCl_6\cdot 6$   $H_2O$  in a stream of chloride. The starting material is placed in a boat set in a

combustion tube made of high-melting glass. The temperature is increased slowly from 60 to 150°C, which drives off water. When the material is completely dry, the temperature is raised to 275-300° over a period of two hours, and held at this level for 0.5 hour. The temperature must not be allowed to rise above 360°C. After cooling to about 150°C, the product is removed, quickly ground, replaced in the combustion boat, and reheated for 0.5 hour at 275°C while a stream of  $Cl_2$  is passed over it. The resultant chloride is placed, while still hot, in a hermetically closable storage bottle. The yield from 6 g of  $H_2PtCl_6 \cdot 6 H_2O$  is 3.7 g. of  $PtCl_4$ .

PROPERTIES:

Red-brown, very hygroscopic crystals. Very soluble in water, sparingly soluble in alcohol. Absorbs moisture on standing in air, yielding  $PtCl_4 \cdot 5 H_2O$ .

**REFERENCES:** 

 A. Gutbier. Z. anorg. Chem. <u>81</u>, 381 (1913); M. S. Kharasch and T. A. Ashford. J. Amer. Chem. Soc. <u>58</u>, 1736 (1936); R. N. Keller in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 247.

PtCl<sub>2</sub>

I. Prepared by heating platinum sponge to about  $500 \,^{\circ}\text{C}$  in a stream of Cl<sub>2</sub> or, better, by thermal decomposition of PtCl<sub>4</sub> or H<sub>2</sub>PtCl<sub>6</sub> · 6 H<sub>2</sub>O.

The  $H_2PtCl_6 \cdot 6 H_2O$  (or the commerical product containing 40% Pt) is subjected to a preliminary decomposition over a free flame at 150 °C. The resulting residue is ground and decomposed in a slow air stream at 360 °C. Depending on the quantities involved, the operation may require several hours. At the end the undecomposed  $H_2PtCl_6$  should be removed by extracting and washing with  $H_2O$ , after which the product is redried at 360 °C.

PROPERTIES:

Formula weight 266.0. Greenish-brown powder. Insoluble in water. At 250°C, dry  $PtCl_2$  forms a very volatile carbonyl chloride with CO. Thus gases containing CO should not be used in reductive ignition of platinum chloride (the same holds for the chlorides of the other platinum metals). Very sparingly soluble in dil. hydrochloric acid, yielding  $H_2$  PtCl<sub>4</sub>.

REFERENCE:

M. S. Kharasch and T. A. Ashford. J. Amer. Chem. Soc. <u>58</u>, 1776 (1936).

II. A chocolate-brown form of platinum (II) chloride can be prepared by careful concentration of a solution of tetrachloroplatinic (II) acid (see p. 1570) in hydrochloric acid; this material is more soluble in hydrochloric acid and aqueous ammonia than the product obtained by method I.

REFERENCE:

W. E. Cooley and D. H. Busch in: T. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 208.

# Hexachloroplatinic (IV) Acid

#### H<sub>2</sub>PtCl<sub>6</sub> · 6 H<sub>2</sub>O

Obtained by dissolving platinum in aqua regia. To prepare large quantities, the platinum, in the form of thin shavings of foil, is dissolved in a porcelain or glass vessel and the solution poured into a porcelain dish. Hydrochloric acid is added and the solution evaporated to sirupy consistency in order to drive off the nitric acid and any  $PtCl_4 \cdot 2$  NOCl which may form. The thick solution is taken up with HCl and the resulting solution reevaporated to a sirup. This is repeated several times. Since the last traces of nitric oxide are very difficult to remove, finely divided Pt may also be dissolved in hydrochloric acid through which  $Cl_2$  is bubbled (or nascent  $Cl_2$  may be generated in the solution itself by carefully adding HClO<sub>3</sub> or  $H_2O_2$  to it).

The concentrated solution is placed in a large tared dish and the percentage of platinum in the chloride is adjusted (usually to 39.5 or 40%) by controlled evaporation of the acid on a burner (check on the decrease in weight).

When the required percentage of platinum is reached, the dish is removed from the burner and allowed to stand until the chloride solution becomes a definite slurry. The solution is then stirred with a thick glass rod until it is completely cool.

SYNONYMS:

Chloroplatinic acid, platinum chloride, platinic chloride.

PROPERTIES:

Formula weight  $(H_2PtCl_6 \cdot 6 H_2O)$  518.0. Commercial "platinum chloride" containing 39.5-40% Pt is not a definite hydrate, but has the composition of  $H_2PtCl_6 \cdot 4.5 H_2O$  and is deep orange. Starting material for the preparation of most platinum compounds.

# Tetrachloroplatinic (II) Acid

## H<sub>2</sub>PtCl<sub>4</sub>

Stable only in solution; prepared by reduction of  $H_2PtCl_8$  with a stoichiometric quantity of  $N_2H_4 \cdot 2$  HCl:

 $2 H_2 PtCl_6 \cdot 6 H_2O + N_2 H_4 \cdot 2 HCl \rightarrow 2 H_2 PtCl_4 + N_2 + 6 HCl + 12 H_2O$ 

Commercial platinum chloride (10 g., 40% Pt) is dissolved in 50 ml. of water in a 150-ml. beaker. Then 1.07 g. of solid  $N_2H_4 \cdot 2$  HCl is added in small portions, so that the solution effervesces each time due to evolution of  $N_2$ . Within five minutes of adding the last of the hydrazine salt, the deep red solution is heated on a steam bath until no further gas evolves; it is then filtered to remove the small quantity of platinum black which may deposit out.

PROPERTIES:

Red solution; leaves a brown deposit of  $PtCl_2$  on careful evaporation. Very stable in hydrochloric acid solution. With an excess of ammonia it forms  $[Pt(NH_3)_4]$   $[PtCl_4]$  or  $[Pt(NH_3)_4]Cl_2$ .

**REFERENCE:** 

W. E. Cooley and D. H. Busch in: T. Moeller, Inorg. Syntheses, Vol. V, New York-Toronto-London, 1957, p. 208.

## Ammonium Hexachloroplatinate (IV)

## (NH<sub>4</sub>)<sub>2</sub>PtCl<sub>6</sub>

A dilute, weakly acidic (HCl) solution of  $H_2PtCl_6$  is prepared and, if that is needed, oxidized with  $H_2O_2$ . It is then reacted with an excess of  $NH_4Cl$  (at least three parts by weight of  $NH_4Cl$  to one of Pt) and slowly evaporated to dryness on a steam bath. The salt crust mixed with the resulting residue is broken up with a glass rod and the solids heated on a steam bath with continuous stirring until the powdery mass no longer gives off the odor of HCl. The dry residue is then carefully moistened with some distilled water, taken up in cold saturated  $NH_4Cl$  solution and filtered. It is washed first with  $NH_4Cl$  solution, then with alcohol. The mother liquor should be completely colorless and show only traces of Pt on reaction with  $H_2S$  and  $SnCl_2$ .

#### SYNONYMS:

Platinic salammoniac, ammonium platinichloride, ammonium chloroplatinate.

PROPERTIES:

Formula weight 443.9. Lemon-yellow octahedra. A color of yellow ochre to brick red instead of lemon yellow indicates the presence of other platinum metals, particularly Pd, Ir and Ru. Greenish-yellow to green indicates that Rh is present. Completely decomposed on ignition in a platinum dish, leaving fine particles of platinum sponge. Very sparingly soluble in  $H_2O$ , less soluble in NH<sub>4</sub>Cl solutions. Solubility (15.5 °C) 0.67 g., (100 °C) 1.25 g./100 ml.  $H_2O$ . Colorless solution in conc. ammonia. Like  $K_2PtCl_6$ , insoluble in alcohol.

## Potassium Hexachloroplatinate (IV)

#### K<sub>2</sub>PtCl<sub>6</sub>

Prepared by adding a solution of KCl to  $H_2PtCl_6$  (the ratio of solid components is 3:1). For complete precipitation of the Pt with KCl or NH<sub>4</sub>Cl, the Pt must be completely oxidized to the +4 state and the solution must be as concentrated as possible; however, the solution should not be so concentrated that it becomes viscous while the product is being formed.

SYNONYM:

Potassium platinichloride.

PROPERTIES:

Formula weight 486.0. Pure yellow crystals; dissolve with difficulty in water; insoluble in alcohol. The color changes in the presence of the other platinum metals in the same way as does that of  $(NH_4)_2PtCl_6$ . Solubility in water:

°C	g. $K_2$ PtCl <sub>6</sub> /100 g. H <sub>2</sub> O
0	0.74
10	0.90
20	1.12
50	2.16
80	3.79
100	5.13

Sodium Hexachloroplatinate (IV)

#### Na<sub>2</sub>PtCl<sub>6</sub>, Na<sub>2</sub>PtCl<sub>6</sub> · 6 H<sub>2</sub>O

Prepared by passing  $Cl_2$  over a mixture of platinum sponge with twice its weight of NaCl. The reaction temperature should be higher

than  $500^{\circ}$ C, but should not under any circumstances exceed  $660^{\circ}$ C. Purification is effected by solution of the product in alcohol, filtration to remove NaCl and platinum metals, and concentration of the solution, after which the salt is dried in a drying oven.

If the reaction mixture is dissolved in water (instead of alcohol) and the solution concentrated,  $Na_2PtCl_6 \cdot 6H_2O$  is produced as triclinic, orange-colored crystals. A very pure salt is obtained by recrystallization from 1% soda solution; it loses its water of crystallization below 100°C.

SYNONYM:

Sodium platinichloride.

PROPERTIES:

Formula weight (anhydrous) 453.8. Orange crystals. Soluble in water and alcohol.

REFERENCE:

L. Wöhler and P. Balz. Z. anorg. allg. Chem. 149, 356 (1925).

# Potassium Tetrachloroplatinate (II)

# K<sub>2</sub>PtCl<sub>4</sub>

Prepared by reduction of  $K_2PtCl_6$  with  $SO_2$  or  $N_2H_4 \cdot 2$  HCl.

I. A suspension of 4.7 g. of K<sub>2</sub>PtCl<sub>6</sub> in 35 ml. of H<sub>2</sub>O is prepared in a 50-ml. beaker; small portions of freshly prepared SO<sub>2</sub> solution are added while stirring the suspension mechanically and heating it to 85-90°C on a water bath. About 15 additions of 0.6 ml. should be made first, followed by 10-15 additions of 0.4 ml. After each addition, 2-3 minutes (later 3-4 minutes) should be allowed until the SO<sub>2</sub> is consumed and its odor disappears. Toward the end of the reduction, when the suspended particles are gradually disappearing, it is necessary to proceed even more slowly. The solution remaining on complete reduction is concentrated on a water bath until crystallization begins. After cooling, it is suction-filtered, and the red K<sub>2</sub>PtCl<sub>4</sub> is dissolved in 40 ml. of cold water, filtered to remove any small residue of  $K_2$  PtCl<sub>e</sub>, and the residue rinsed with 5 ml. of water. The solution is carefully transferred (rinsing with 10 ml. of water) into an 800-ml. beaker; its total volume at this point is 55-60 ml. Then 660 ml. of a 1:1 mixture of acetone and ether is added with stirring. This precipitates the solid chloroplatinate,

1572

which is allowed to settle. The bright yellow liquid is decanted, and the salt is washed three times with 120 ml. of acetone-ether mixture (decantation) and then three times with 80 ml. of ether. After filtering and drying the product in air, the salt is recrystallized from hot, slightly acidified water.

II. The  $K_2PtCl_6$  can be reduced with  $N_2H_4 \cdot 2$  HCl in the same way as  $H_2PtCl_6$ .

 $2 K_2 PtCl_6 + N_2 H_4 \cdot 2 HCl = 2 K_2 PtCl_4 + 6 HCl + N_2$ 972.1 105.0 830.2

The  $K_2PtCl_6$  is suspended (brisk stirring) in 10-12 times its weight of water. With continuous stirring, the stoichiometric quantity of solid  $N_2H_4 \cdot 2$  HCl is added in small portions, the temperature being raised to 50 °C within 10 minutes. The  $K_2PtCl_6$ dissolves with evolution of  $N_2$ . The solution is brought to the boiling point, filtered and concentrated, first over an open flame and then on a water bath, until crystallization begins. Since no foreign ions are introduced in this preparative method, recrystallization can be omitted, provided the starting materials used are pure.

An excess of  $N_2H_4 \cdot 2$  HCl leads to the formation of platinum black, while a deficiency leaves undissolved  $K_2PtCl_6$ .

The compounds  $(NH_4)_2$ PtCl<sub>4</sub> and Na<sub>2</sub>PtCl<sub>4</sub> can also be prepared in this way.

PROPERTIES:

Formula weight 415.1. Red crystals or bright red powder. Solubility in water  $(16^{\circ})$  0.93 g.,  $(100^{\circ}C)$  5.3 g./100 ml. Insoluble in alcohol; reduced in alcohol.

REFERENCES:

- I. Magnus. Pogg. Ann. 14, 241 (1828).
- II. R. N. Keller in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 247; N. G. Klynchnikov and R. N. Savel'eva, abstract in Chem. Abs. <u>51</u>, 10288 (1957).

# Platinum (II) Oxide

#### PtO

Wöhler's method is recommended in preference to others. Oxygen is passed at 150°C over platinum sponge that is as finely divided as possible; the temperature must not exceed 560°C.
PROPERTIES:

Completely anhydrous black powder. Readily soluble in aqua regia. Instantly reduced by  $H_2$  at ordinary temperatures, liberating heat and forming gray platinum sponge. Decomposes at atmospheric pressure (560°C): 2 PtO = 2 Pt + O<sub>2</sub>.

**REFERENCE:** 

L. Wöhler. Z. Elektrochem. 15, 136 (1909).

# Platinum (IV) Oxide

#### PtO<sub>2</sub> · xH<sub>2</sub>O

I. It is impossible to prepare completely anhydrous  $PtO_2$  without decomposing it. In addition, even the hydrated material (which contains variable quantities of water) is not easy to obtain in pure form. The procedure of Wöhler and Frey appears to be the most suitable. Red-brown hydrated platinum dioxide is precipitated by boiling a pure, concentrated solution of  $H_2PtCl_6$  with concentrated  $Na_2CO_3$  solution. The precipitated solid is made acid-insoluble by heating for several hours in a drying oven at 200°C; it is then freed of chlorides by vigorous boiling with dilute soda solution and distilled water. Finally, it is freed of alkali by treatment with dilute sulfuric acid and distilled water, filtered off and dried on a water bath.

PROPERTIES:

Straw yellow after brief drying; on further drying, becomes yellow ochre and then dark brown; at this point the oxide is acidinsoluble.

REFERENCE:

L. Wöhler. Z. anorg. Chem. 40, 436 (1904).

II. A 10% solution of  $H_2PtCl_6$  (10 ml.) is added to 9 g. of NaNO<sub>3</sub> in a 50-ml. beaker. The solution is evaporated to dryness over a flame (covered with asbestos-wire gauze) while continuously stirring with a glass rod. Local melting of the mixture must be carefully avoided. Then 100 g. of NaNO<sub>3</sub> is heated to 520 °C in a 400-ml. beaker (controlling the temperature with a thermocouple), and the dry, fine powder residue from the evaporation is added at once. The flame is then removed. The olive-brown platinum oxide is

precipitated more or less quantitatively and settles rapidly. During cooling and solidification, the beaker is rotated, so that the contents solidify on the walls in fine particles, thus avoiding the cracking of the beaker.

The cooled melt is dissolved in about two liters of  $H_2O$ , and the residue is filtered off with suction. It is then washed repeatedly with  $H_2O$ , taking care that the residue remains covered with water at all times, since otherwise it passes into solution as a colloid. Finally it is dried over CaCl<sub>2</sub> in an evacuated desiccator.

PROPERTIES:

Heavy brown powder; insoluble in aqua regia. Blackens and settles rapidly when treated with  $H_2$  in an alcoholic suspension (this powder is soluble in aqua regia). Very active catalyst for the hydrogenation of olefins and carbonyl groups.

**REFERENCE:** 

V. L. Framton, J. D. Edwards, Jr., and H. R. Henze. J. Amer. Chem. Soc. <u>73</u>, 4432 (1951).

## Hexahydroxyplatinates (IV)

# $Na_2Pt(OH)_6 \cdot x H_2O, K_2Pt(OH)_6 \cdot x H_2O$

An aqueous solution of  $Na_2PtCl_8$  or  $K_2PtCl_8$  is boiled with NaOH and then treated with alcohol. The precipitate consists of small colorless crystals, which are filtered off and dried in air. Depending on the conditions of precipitation, the product contains from 0.5 to 3 moles of H<sub>2</sub>O. Aqueous solutions of these two salts are very good electrolytes for electroplating platinum.

# Platinum (II) Sulfide

PtS

Produced by heating an intimate mixture of very fine powders of platinum sponge and sulfur; may also be produced by decomposition of a boiling solution of  $PtCl_2$  by bubbling in  $H_2S$ , or by adding to it a solution of one of the alkali sulfides. The black precipitate can be washed and dried without being altered.

PROPERTIES:

Insoluble in acids, even at the boil. Yields metallic Pt when heated in air.

# Platinum (IV) Sulfide

## PtS<sub>2</sub>

Produced as a dark-brown precipitate by passing  $H_2S$  through a hot solution of  $H_2PtCl_8$  in hydrochloric acid. The precipitation can be greatly accelerated and made almost quantitative if the weakly acidic platinate solution is mixed with a 5% solution of MgCl<sub>2</sub> and then saturated with  $H_2S$  gas. After the excess  $H_2S$  is driven off by boiling, the PtS<sub>2</sub> is filtered off and carefully dried.

PROPERTIES:

Insoluble in hydrochloric and sulfuric acids, soluble in nitric acid and particularly in aqua regia. Even though Pt belongs to the group of elements forming this salts,  $PtS_2$  is only slightly soluble in colorless alkali sulfides and yellow ammonium sulfide.

# Potassium Tetracyanoplatinate (II) and Barium Tetracyanoplatinate (II) K<sub>2</sub>Pt(CN)<sub>4</sub>·3 H<sub>2</sub>O, BaPt(CN)<sub>4</sub>·4 H<sub>2</sub>O

Both salts are obtained by precipitation reactions.

PtCl <sub>2</sub>	(K2PtCl4)	+ 4	KCN	=	$K_2Pt(CN)_4$	+	2 KCl	(4 KCl)
266.0	(415.1)		260.5		(3H <sub>2</sub> O) 431.4		149.1	(298.2)
	K <sub>2</sub> Pt(CN) <sub>4</sub>	+ H	BaCl <sub>2</sub>	=	BaPt(CN)4	+	2 KCl	
	(3H <sub>2</sub> O) 431.4	(	2H₂O) 244.3		(5 H <sub>2</sub> O) 526.6		149,1	

A solution of  $PtCl_2$  or  $K_2PtCl_4$  is added to a cold, saturated solution of KCN (use a good hood!). The precipitated  $K_2Pt(CN)_4 \cdot$  $3 H_2O$  is filtered off with suction. If it is to be used for preparing the barium salt, it is dissolved in water and treated with a concentrated aqueous solution of  $BaCl_2$ . The precipitated  $BaPt(CN)_4 \cdot$  $4 H_2O$  is filtered off with suction and washed with cold water.

SYNONYMS:

Potassium platinocyanide; barium platinocyanide.

PROPERTIES:

 $K_2Pt(CN)_4 \cdot 3 H_2O$ : Polychromatic, blue and yellow. Readily soluble in hot water; most of it rapidly reprecipitates on cooling the solution. d 2.455.

Ba[Pt(CN)<sub>4</sub>] · 4 H<sub>2</sub>O: Crystals with brilliant polychromism (pleochroism), iridescent violet-blue on the prism faces, yellowgreen in the axial direction. Solubility (20°C) 3.5 g./100 g. H<sub>2</sub>O. d 2.076.

# Ammine Complexes of Platinum (II)

# (Platinum Ammines)

# Magnus's Salt [Pt(NH<sub>3</sub>)<sub>4</sub>][PtCl<sub>4</sub>] and Reiset's First Chloride [Pt(NH<sub>3</sub>)<sub>4</sub>]Cl<sub>2</sub>·H<sub>2</sub>O

An excess of 50% ammonia is added to a boiling solution of  $H_2PtCl_4$  (obtained by reduction of  $H_2PtCl_6$ , see p. 1570). Cooling precipitates the dark-green crystals of Magnus's salt. If heating with the excess of ammonia is continued with stirring, ignoring the appearance of the precipitate, the latter redissolves and the solution becomes colorless. It now contains  $[Pt(NH_3)_4]Cl_2$ , Reiset's first chloride. This salt can be precipitated directly by adding alcohol whereby it is obtained as colorless crystals. However, for higher purity, the preparation should proceed via Magnus's salt.

This requires a solution of  $H_2PtCl_4$  in which the latter is present in a quantity exactly equivalent to the Reiset's chloride. The simplest way to achieve this is to divide a given quantity of  $H_2PtCl_4$  solution into two equal parts and use only one of these for conversion of the solute to  $[Pt(NH_3)_4]Cl_2$  in the manner described above. The excess  $NH_3$  is driven off as completely as possible by heating on a water bath. The two solutions are then gradually combined (stirring), and pure green Magnus's salt is precipitated:

 $[Pt(NH_3)_4]Cl_2 + H_2PtCl_4 = [Pt(NH_3)_4][PtCl_4] + 2 HCl_4$ 

The precipitate is allowed to settle, the mother liquor decanted, and the solid washed with small portions of hot water (on a filter) until the wash water is free of chlorides. The Magnus's salt is now pure and can be dried.

For conversion into pure Reiset's chloride, the moist Magnus's salt is placed in a beaker, covered with some dil. hydrochloric acid, and treated with an excess of concentrated ammonia. The mixture is boiled gently with continuous stirring, gradually dissolving the solids. The evaporating ammonia must be replaced from time to time to maintain the original volume. After the green salt is completely dissolved the solution is evaporated until only a faint odor of  $NH_3$  remains. Then it is neutralized to litmus, 1 ml. of concentrated HCl added, and the mixture treated with 10 times its volume of 1:1 alcohol-acetone. It is allowed to stand for an

hour; the white precipitate is removed, washed a few times with small portions of alcohol-acetone, and rinsed with pure acetone on a suction filter. The resulting Reiset's chloride is dried in air.

PROPERTIES:

Magnus's salt: Dark-green crystalline needles. Very difficult to dissolve in water. Rapidly transformed into trans- $[PtCl_2(NH_3)_2]$  on dry heating to 290 °C.

Reiset's first chloride: Colorless tetragonal crystals. Solubility (20 °C) about 20 g./100 g.  $H_2O$ ; more soluble in hot water. Insoluble in alcohol, ether and acetone. Forms the hydrate  $[Pt(NH_3)_4]Cl_2 \cdot H_2O$  on recrystallization or concentration of an aqueous solution.

# **Reiset's Second Chloride**

#### trans-[PtCl<sub>2</sub>(NH<sub>3</sub>)<sub>2</sub>]

If Reiset's first chloride is heated at 250 °C until no further NH<sub>3</sub> is given off, the product is trans-diamminedichloroplatinum (II):

 $[Pt(NH_3)_4]Cl_2 = [PtCl_2(NH_3)_2] + 2 NH_3$ 334.1 300.1 34.1

A mixture of this chloride with  $NH_4Cl$  is obtained by evaporating the first chloride with a large excess of concentrated HCl; the  $NH_4Cl$  is extracted from the residue with cold water.

Purification is effected by recrystallization from hot water or by converting the solid to the nitrate by means of  $AgNO_3$ , followed by reprecipitation of the chloride from the nitrate solution with concentrated HCl.

PROPERTIES:

Sulfur-yellow crystalline powder. Very slightly soluble in cold water; solubility (100°C) 0.7 g./100 g.  $H_2O$ . Decomposes above 340°C.

# Peyrone's Chloride

# cis-[PtCl<sub>2</sub>(NH<sub>3</sub>)<sub>2</sub>]

I. A cold, clear solution of 20 g. of  $(NH_4)_2PtCl_4$  in 100 ml. of  $H_2O$  is reacted with 50 ml. of 5 N ammonia and allowed to stand for

12-48 hours in a closed flask at 0°C. The crystalline precipitate contains Peyrone's chloride and some Magnus's salt. The mixture is filtered off and washed with ice water until no Magnus's salt is precipitated from the filtrate with  $PtCl_4^{2-}$ . The Peyrone's chloride on the filter is then dissolved with boiling water, and the yellow solution is mixed with 1/3 its volume of 50% hydrochloric acid. After standing for 24 hours, the crystalline precipitate is filtered off, washed until free of acid with ice water and then with alcohol, and finally dried in air. Yield: 10.7 g.

II. A lukewarm solution of 41 g. of  $K_2PtCl_4$  and 27 g. of  $NH_4Cl$  in 200 ml. of  $H_2O$  is reacted with 54.4 ml. of 3.75 N ammonia (0.204 mole) and allowed to stand for two days at room temperature and for an additional day at 0°C. The precipitate which has formed is then filtered off with suction (removing the liquid as completely as possible), thoroughly washed with ice water, and dried in air. The yield is 27.1 g. of a product which is not completely pure but contains, in addition to cis-[PtCl<sub>2</sub>(NH<sub>3</sub>)<sub>2</sub>], a few percent of Magnus's salt and [PtCl(NH<sub>3</sub>)<sub>3</sub>]<sub>2</sub> [PtCl<sub>4</sub>].

PROPERTIES:

Yellow crystalline powder (needles or platelets). Solubility  $(0^{\circ}C) 0.26 \text{ g.}$ ,  $(100^{\circ}C) 3 \text{ g.}/100 \text{ ml. H}_{2}O$ . Dissolves very slowly in water at  $100^{\circ}C$ . Rapidly converted to the trans compound on dry heating to  $275^{\circ}C$ .

**REFERENCES:** 

Gmelins Handb. d. anorg. Chem. [Gmelin's Handbook of Inorganic Chemistry], 8th ed. (1957), Platinum, Part D, pp. 45, 53, 236, 241; S. M. Jörgensen. Z. anorg. Chem. <u>24</u>, 153 (1900); S. M. Jörgensen and S. P. L. Jörgensen. Ibid. <u>48</u>, 441 (1906); L. Ramberg. Ibid. <u>83</u>, 33 (1913); R. N. Keller in: W. C. Fernelius, Inorg. Syntheses, Vol. II, New York-London, 1946, p. 250.

# cis-Dinitrodiammineplatinum (II)

# [Pt(NO<sub>2</sub>)<sub>2</sub>(NH<sub>3</sub>)<sub>2</sub>]

The cis form of this neutral salt complex is precipitated when an aqueous solution of potassium platinum (II) nitrite is treated with aqueous ammonia.

The starting  $K_2Pt(NO_2)_4$  can be prepared from  $K_2PtCl_6$ , which is allowed to react with an excess of alkali nitrite, evolving nitric oxide.

One part by weight of  $K_2PtCl_6$  is suspended in water and treated with a concentrated solution of 10 parts by weight of NaNO<sub>2</sub>. The mixture is then heated with stirring. The yellow  $K_2PtCl_6$  first dissolves giving a dark solution, and then nitric oxide is liberated as fine bubbles while the solution clears to a pale greenish yellow. When no further gas evolves the solution is cooled and, if necessary, the precipitated impurities are filtered off.

The  $K_2$ Pt(NO<sub>2</sub>)<sub>4</sub> solution can also be prepared, in a smooth reaction, from  $K_2$ PtCl<sub>4</sub>:

$$\begin{array}{rrrr} K_2 PtCl_4 &+ & 4 NaNO_2 &= & K_2 Pt(NO_2)_4 &+ & 4 NaCl_4 \\ 415.1 & & 276.0 & & 457.3 & & 233.8 \end{array}$$

To produce the desired complex  $[Pt(NO_2)_2(NH_3)_2]$ , the cold, filtered solution of  $K_2Pt(NO_2)_4$  is reacted with a stoichiometric quantity of 20% aqueous ammonia:

 $\begin{array}{rrrr} K_2 Pt(NO_2)_4 &+& 2 NH_8 &=& [Pt(NO_2)_2(NH_3)_2] &+& 2 KNO_2 \\ 457.4 & 34.1 & 321.2 & 170.2 \end{array}$ 

After a short time the complex precipitates as a dense whitish mass of fine, needlelike crystals. After filtration and washing with cold water, it can be recrystallized from hot water; the product consists of pale-yellow needles.

PROPERTIES:

Formula weight 321.17. Pale-yellow needlelike crystals. Decompose explosively at 200°C. Sparingly soluble in water; readily soluble in aqueous ammonia, forming  $[PtNO_2(NH_3)_3]NO_2$ , which can be used to prepare a good platinum electroplating bath.

REFERENCE:

W. Keitel and H. E. Zschiegner. U.S. Patent 1,779,436.

# Pure Palladium

#### Pd

In the Wilm method, very pure palladium is obtained by treating a solution of  $PdCl_2$  or  $Na_2PdCl_4$  with  $NH_4Cl$  in order to precipitate as  $(NH_4)_2PtCl_6$  any slight Pt impurity which may be present. The filtrate is boiled with an excess of  $NH_3$ , filtered again if necessary, and acidified with HCl. A yellow precipitate of very pure  $[PdCl_2(NH_3)_2]$  should form. If the salt has a dull, dirty yellow color,

it contains a small quantity of  $[RhCl(NH_3)_8]Cl_2$ , which is insoluble in cold ammonia. The salt is therefore digested with cold aqueous ammonia; completely pure palladodiammine chloride  $[PdCl_2(NH_3)_2]$ is obtained from the filtrate by a second precipitation with hydrochloric acid; it is a bright-yellow crystalline salt. This is reduced by ignition in a stream of H<sub>2</sub> to light gray palladium sponge.

PROPERTIES:

M.p. 1554°C; d 11.97. Absorbs large quantities of many gases, especially  $H_2$ .

REFERENCE:

Th. Wilm. Ber. dtsch. chem. Ges. 15, 241 (1882).

# Colloidal Palladium

A solution of 2 g. of sodium protabinate (the sodium salt of protabinic or lysalbinic acid) in 50 ml. of water is prepared; aqueous NaOH is added in slight excess, followed by a solution of 1.6 g of PdCl<sub>2</sub> (equivalent to 1 g. of Pd) in 25 ml. of H<sub>2</sub>O. Then N<sub>2</sub>H<sub>4</sub>  $\cdot$  H<sub>2</sub>O is added dropwise to the tesulting clear red-brown liquid, producing immediate reduction (foaming). After standing for three hours, the black solution is dialyzed against water to remove the excess NaOH, N<sub>2</sub>H<sub>4</sub>  $\cdot$  H<sub>2</sub>O and NaCl; this is continued until the dialyzing water no longer gives a reaction for N<sub>2</sub>H<sub>4</sub>  $\cdot$  H<sub>2</sub>O and NaCl. The purified solution is concentrated at 60-70°C and dried in vacuum over H<sub>2</sub>SO<sub>4</sub>. The product consists of shiny black platelets, which dissolve in water leaving no residue.

PROPERTIES:

Stable when dry. Its solution appears opaque and black in incident light; thin layers are clear black-brown with a greenish tinge in transmitted light. One volume of the product (about 50% Pd) contained in this colloidal solution can absorb approximately 3000 volumes of  $H_2$ .

**REFERENCES:** 

C. Paal and C. Amberger. Ber. dtsch. chem. Ges. <u>37</u>, 124 (1904); P. Stecher et al. Merck Index, 7th ed., p. 623 under "lysalbinic acid."

# Palladium Black

The Böttger method for preparing palladium black consists in reducing an aqueous solution of a Pd (II) salt with sodium formate.

The reaction occurs slowly at room temperature and is instantaneous at 50  $^{\circ}\mathrm{C}$  .

PROPERTIES:

According to C. Paal, an aqueous suspension of palladium black absorbs 12,000 times its volume of  $H_2$ ; the dry material absorbs only 870 times its volume.

REFERENCE:

Jahresber. d. phys. Vereins Frankfurt a.M. [Annual Report of the Frankfurt a.M. Physics Society], 1872-73, p. 11.

# **Palladized Asbestos**

Palladized asbestos is prepared in exactly the same way as platinized asbestos (see p. 1563).

# Palladium (II) Cloride

#### PdCl<sub>2</sub>

The anhydrous salt is prepared by heating loose palladium sponge (contained in a porcelain boat set in a glass tube) to a dull red heat in a stream of  $Cl_2$ . According to Krustinsons, the decomposition pressure of  $PdCl_2$  reaches 1 atm. at 738 °C.

By dissolving finely divided Pd in conc. HCl through which  $Cl_2$ is bubbled, one obtains a solution in which both  $H_2PdCl_4$  and  $H_2PdCl_6$  can be detected. Concentrating the solution also yields a residue of PdCl<sub>2</sub>.

# PdCl<sub>2</sub> Solution for the Detection of CO

Winkler gives the following method for preparing this solution. Pure Pd (0.2 g.) is dissolved with gentle heating in about 10 ml. of aqua regia. The solution is evaporated to dryness in a 50-ml. porcelain dish placed on a steam bath. The residue is dissolved in 10 ml. of 20% hydrochloric acid and the solution is again evaporated to dryness; this last procedure is repeated three times. The resulting residue, which is now completely nitrate-free, is mixed with 2 g. of KBr and dissolved (gentle heating) in 10 ml. of 1 N HCl. After dilution to about 150 ml. with water, a few particles of pumice and 1 ml. of alcohol are added to the solution, which is then boiled for about 10 minutes in an Erlenmeyer flask in order

to reduce any Pd (IV) not decomposed during the drying to Pd (II) and to drive off the excess alcohol. After cooling, 2.5 g. of  $CH_3COONa \cdot 3 H_2O$  is dissolved in the liquid. The solution is filtered through a small wad of cotton wool and diluted to 200 ml. with the water used for washing the cotton wool. The clear, reddish-brown liquid, which contains 0.1% palladium, is stable when stored in a flask provided with a ground-glass stopper. To be on the safe side, it is best to filter the solution before use; it should be stored in the dark.

**REFERENCES:** 

J. Krustinsons. Z. Elektrochem. <u>44</u>, 537 (1938); L. Winkler. Z. anal. Chem. <u>100</u>, 321 (1935); <u>97</u>, 18 (1934); also describes analytical methods for detecting CO with PdCl<sub>2</sub> solutions. Explicit directions for the preparation of palladium catalysts using PdCl<sub>2</sub> are given by R. Mozingo in Organic Syntheses, collective vol. III, p. 685 (Wiley, New York, 1955).

# Palladium (II) Oxide

#### PdO

A reasonably pure PdO, particularly suitable for catalytic purposes, can be prepared by decomposition of palladium nitrate. Sodium nitrate (50 g.) and a solution of PdCl<sub>2</sub> containing 2 g. of Pd are mixed and evaporated to dryness. The dry mixture is then heated (it fuses in the process), first for some time at 270-280°C, then at 350-370°C, until evolution of nitric oxides ceases; finally, it is heated to 575-600°C for a short time. The melt is extracted with 200 ml. of water, leaving behind the PdO. This is washed with a 1% NaNO<sub>3</sub> solution and dried in vacuum over H<sub>2</sub>SO<sub>4</sub>. The product still contains about 1.5% H<sub>2</sub>O and 2.5% alkali salts. The pure material can be obtained by ignition in O<sub>2</sub>, but this causes a loss of catalytic activity.

PROPERTIES:

Black powder; tetragonal crystals. Stable in air up to about 700°C, in  $O_2$  to about 800°C. Insoluble in aqua regia; soluble in conc. HBr. d 8.7.

#### **REFERENCES:**

Gmelin. Handb. d. anorg. Chem. [Gmelin's Handbook of Inorganic Chemistry], 8th ed., System No. 65, Berlin, 1942; R. L. Shriner and R. Adams. J. Amer. Chem. Soc. <u>46</u>, 1685 (1924).

#### H. L. GRUBE

# Tetrachloropalladates (II)

# K<sub>2</sub>PdCl<sub>4</sub>, Na<sub>2</sub>PdCl<sub>4</sub>, (NH<sub>4</sub>)<sub>2</sub>PdCl<sub>4</sub>

These three salts are obtained as well-formed crystals by treating  $PdCl_3$  solutions with stoichiometric quantities of the respective alkali chlorides and slowly evaporating the solutions.

SYNONYMS:

Potassium, sodium and ammonium palladochlorides.

PROPERTIES:

 $K_2PdCl_4$ : Crystallizes in dark yellow or brownish prisms. Readily soluble in hot water, soluble with difficulty in cold water. Precipitated in golden yellow lamellae by addition of alcohol to a hot aqueous solution.

Na<sub>2</sub>PdCl<sub>4</sub>: Brown, deliquescent; also soluble in alcohol.

(NH<sub>4</sub>)<sub>2</sub>PdCl<sub>4</sub>: Crystallizes in long olive-colored prisms; can be recrystallized from water.

# Hexachloropalladates (IV)

## K<sub>2</sub>PdCl<sub>6</sub>, (NH<sub>4</sub>)<sub>2</sub>PdCl<sub>6</sub>

A solution of  $PdCl_2$  with an excess of KCl (NH<sub>4</sub>Cl) is prepared, from which bright red  $K_2PdCl_6$  [(NH<sub>4</sub>)<sub>2</sub>PdCl<sub>6</sub>] is precipitated on introduction of chlorine. This is rapidly suction-filtered, washed quickly with KCl (NH<sub>4</sub>Cl)-containing water, and rinsed with alcohol.

SYNONYMS:

Potassium and ammonium palladium (IV) chlorides.

PROPERTIES:

Bright red crystals. Soluble with difficulty in water, even less soluble in KCl and  $NH_4$  Cl solutions. Crystal structure:  $K_2PtCl_8$  type.

# Diamminepalladium (II) Salts

## $[PdCl_2(NH_3)_2], [PdBr_2(NH_3)_2]$

If a slight excess of ammonia is added to a fairly dilute, cold solution of  $PdCl_2$ , a red precipitate known as Vauquelin's salt is formed. After drying, this becomes a crystalline, flesh-colored to dark-red powder, corresponding to the formula  $[Pd(NH_3)_4]$ - $[PdCl_4]$  (analogous to Magnus's green platinum salt). On boiling in water, most of it dissolves; the solution precipitates small yellow octahedral crystals on cooling; this is trans- $[PdCl_2(NH_3)_2]$ . Larger quantities can be easily prepared via methods described in the section on the preparation of pure palladium (p. 1580).

A PdBr<sub>2</sub> solution behaves in an exactly parallel manner upon addition of ammonia: a red, crystalline precipitate of  $[Pd(NH_3)_4]$ - $[PdBr_4]$  is obtained from the mixture. This undergoes the same transformation as the chloride to give yellow octahedral crystals of  $[PdBr_2(NH_3)_2]$ .

## Pure Rhodium

#### Rh

I. Reasonably pure sodium or potassium hexachlororhodate (see p. 1588 for preparation) is the starting material. The salt is dissolved in water; the solution is boiled with an excess of ammonia and concentrated. This gives the so called purpureo salt [RhCl(NH<sub>3</sub>)<sub>5</sub>]Cl<sub>2</sub> as a straw-colored powder, which must be purified. The salt is first digested for a long time in hot 50% hydrochloric acid, is then suction filtered (removing as much water as possible) and dried. The lumps are carefully broken up with a broad glass spatula and transferred to a container of cold, concentrated  $H_2SO_4$  (salt:  $H_2SO_4$  ratio = 1:1.5). Too large an excess of  $H_2SO_4$  should be avoided, and the mixture should be warmed very carefully, since otherwise an insoluble sulfate will result. Upon addition of the powder, small lumps, not wetted by the  $H_2SO_4$ , are easily formed; these must be broken up with the glass spatula. The powder must be added in small portions with continuous stirring; the HCl escapes in bubbles, so that the mass foams. It is digested until it becomes a honeylike, viscous, lump-free yellowish paste. Hot water is added and the mass is filtered; the filtrate is allowed to run into concentrated HCl so that the resulting solution is approximately 50% in hydrochloric acid. The purified purpureo salt precipitates as a dense, yellowish-white residue. It is filtered off with suction, washed, dried, ground with a glass spatula, and boiled

for two hours with five times its quantity of concentrated HNO<sub>3</sub>, after which the solution is mixed with an equal volume of water. The nitrate is allowed to crystallize overnight; it is then filtered off, washed and recrystallized once from water. It is then redissolved in water, the solution is filtered, and the filtrate is again allowed to run into hydrochloric acid. The salt is washed with liquids whose compositions approximate those of the respective mother liquors. Since the salt is only slightly soluble in cold  $H_2O$ , it is given a final rapid wash with cold  $H_2O$ , preferably on a filter connected to a vacuum pump. The purpureochloride obtained in this way is placed in a covered quartz crucible set inside a graphite crucible and ignited carefully in a gas-heated or muffle furnace.

II. According to Wichers and Gilchrist, pure rhodium can be prepared as follows. The finely divided, impure metallic raw material is mixed intimately with 1.5 times its weight of NaCl and heated at 600°C in a stream of Cl<sub>2</sub> for 2-4 hours. It is then cooled in the stream of  $Cl_2$ , and the fused mass is dissolved in  $H_2O_2$ . The insoluble residue is again treated with chlorine until all of the rhodium becomes soluble. The solution is then diluted to a concentration of 40 g. of Rh/liter and filtered. The filtrate is heated on a steam bath and NaNO<sub>2</sub> is added until the color changes from red to yellow; this requires about 500-550 g. of NaNO<sub>2</sub> per 100 g. of Rh. Finally the solution is boiled for an hour. The platinum metals and some of the base metals are converted into soluble double nitrites, while other base metals are precipitated as hydroxides or basic salts. The mixture is filtered; the cold solution is treated with Na<sub>2</sub>S and allowed to stand overnight (5-10 g. of Na<sub>2</sub>S is sufficient for a solution containing several hundred g. of Rh). The odor of H<sub>2</sub>S indicates the end of the reaction, which precipitates Pb and small quantities of Pd, Pt and Ir. The filtrate is boiled to decompose the excess Na<sub>2</sub>S. The purified solution is again treated with 30-50 g. of NaNO2 per 100 g. of Rh (to convert the rhodium completely to the double nitrite). The cooled solution is treated with a saturated solution of NH<sub>4</sub>Cl, which precipitates the sparingly soluble  $(NH_4)_3[Rh(NO_2)_6]$ , which is white when pure. This product is allowed to react with hydrochloric acid. The resulting hydrochloric acid solution of rhodium chloride is treated with NaNO<sub>2</sub> (after evaporating the excess of the acid) and treated again as described above, except that smaller additions of Na<sub>2</sub>S are made in the successive purifications. Finally, the concentrated solution of rhodium chloride in hydrochloric acid is converted to (NH<sub>4</sub>)<sub>3</sub>RhCl<sub>6</sub> · H<sub>2</sub>O by addition of a small excess of NH<sub>4</sub>Cl, and the mixture is treated with 95% alcohol. The precipitate is filtered off and washed with alcohol. The (NH<sub>4</sub>)<sub>3</sub>RhCl<sub>6</sub> · H<sub>2</sub>O may be redissolved in water and reprecipitated with alcohol.

The  $(NH_4)_3RhCl_6 \cdot H_2O$  is ignited to rhodium sponge and post-reduced with hydrogen.

**PROPERTIES:** 

M.p. 1970 °C. Harder and more difficult to work than Pt. The solid metal and the fine rhodium black powder obtained by reduction from salt solutions differ in their solubility in acids. The solid metal is insoluble in all acids and mixtures of acids, and is not attacked by molten NaOH even if  $KNO_3$  is added at dull red heat. If Rh is fused with  $KHSO_4$ , it slowly forms the water-soluble potassium rhodium sulfate, which imparts a dark-red color to the melt; at high Rh concentrations, the melt becomes black.

**REFERENCE:** 

E. Wichers and R. Gilchrist, Trans. Amer. Inst. Mining Metallurg. Eng. <u>76</u>, 619 (1928).

# Rhodium (III) Chloride

#### RhCl<sub>3</sub>

The anhydrous chloride is prepared by heating the metal in a stream of  $Cl_2$  at about 400°C. Above 800°C, it redecomposes to the metal and chlorine. This chloride is red and insoluble in water and acids.

However, the hydrated rhodium (III) oxide mentioned on page 1588 dissolves readily in hydrochloric acid, giving a yellow solution. On evaporation of this solution, a residue of the hydrated chloride  $RhCl_3 \cdot xH_2O$  (x = 3-4) is left as a red deliquescent mass, which is called "water-soluble rhodium chloride" to distinguish it from the first product. Heating above 200°C converts this product to the water-insoluble  $RhCl_3$ .

## Hexachlororhodates (III)

# Sodium hexachlororhodate (III), Na<sub>3</sub>RhCl<sub>6</sub>·12H<sub>2</sub>O

First RhCl<sub>3</sub> is prepared by passing Cl<sub>2</sub> over very fine rhodium powder at about 400°C. One part by weight of the product RhCl<sub>3</sub> is carefully mixed with 2-3 parts by weight of NaCl and heated to about 300°C in a stream of Cl<sub>2</sub>. The aqueous solution of this chlorination product is filtered; after concentration, Na<sub>3</sub>RhCl<sub>6</sub>. 12 H<sub>2</sub>O crystallizes out as deep-red, monoclinic prismatic crystals.

# Potassium hexachlororhodate (III), K<sub>3</sub>RhCl<sub>6</sub>·H<sub>2</sub>O, K<sub>2</sub>[RhCl<sub>5</sub>(H<sub>2</sub>O)]

A solution of the potassium salt is prepared in exactly the same way as that of the sodium salt. On concentration, the first crystallization yields  $K_2[RhCl_5(H_2O)]$ . This salt is dissolved in an almost saturated aqueous KCl solution and the solution concentrated; the hexachlororhodate  $K_3RhCl_6 \cdot H_2O$  crystallizes out on cooling. Both of the above compounds form dark-red crystals.

# Ammonium hexachlororhodate (III), (NH<sub>4</sub>)<sub>3</sub>RhCl<sub>5</sub>·H<sub>2</sub>O, (NH<sub>4</sub>)<sub>2</sub>[RhCl<sub>5</sub>(H<sub>2</sub>O)]

Concentration of a platinum-rhodium solution which has been freed of platinum by the addition of  $NH_4Cl$  yields crystals of the red  $(NH_4)_3RhCl_6 \cdot H_2O$ . Green crystals occasionally obtained are  $(NH_4)_2PtCl_6$  containing Rh as an impurity.

A better method starts with soluble rhodium chloride (see page 1587), which is evaporated together with an excess of aqueous  $NH_4Cl$ .

If the  $(NH_4)_3 RhCl_6 \cdot H_2O$  is taken up in water and heated to a high temperature, the much less soluble  $(NH_4)_2[RhCl_5(H_2O)]$  crystallizes out on cooling.

REFERENCE:

M. Delépine. Bull. Soc. Chim. Belgique 36, 114 (1927).

# Rhodium (III) Oxide

## Rh<sub>2</sub>O<sub>3</sub>

I. Very pure  $Rh_2O_3$  is obtained by heating  $RhCl_3$  to 750-800°C in a stream of  $O_2$  until  $Cl_2$  is no longer given off.

II. The highly hydrated compound  $Rh_2O_3 \cdot 5 H_2O$  is obtained when concentrated KOH is added slowly to solutions of rhodium salts. A lemon-yellow compound precipitates; after washing and drying, it becomes a pale-yellow powder. This material is not completely alkali-free and is insoluble in water; however, it dissolves readily in acids and on ignition reverts to nonhydrated  $Rh_2O_3$ , which is insoluble in acids.

SYNONYM:

 $Rh_2O_3 \cdot 5 H_2O$ : Rhodium hydroxide.

**REFERENCES:** 

- I. Gmelin. Handb. d. anorg. Chem. [Gmelin's Handbook of Inorganic Chemistry], 8th ed., Rhodium, p. 46; L. Wöhler and W. Müller. Z. anorg. allg. Chem. 149, 132 (1925).
- II. F. Krauss and H. Umbach. Z. anorg. allg. Chem. <u>180</u>, 47 (1929); G. Grube and G. Bau-Tschang Gu. Z. Elektrochem. <u>43</u>, 398 (1937).

## **Rhodium Sulfate**

 $Rh_2(SO_4)_3 \cdot x H_2O$ 

According to Krauss and Umbach, attempts to prepare rhodium sulfate from rhodium hydroxide and sulfuric acid lead to two different products, depending on the conditions: these are yellow rhodium sulfate  $Rh_2(SO_4)_3 \cdot 15 H_2O$  and red rhodium sulfate  $Rh_2(SO_4)_3 \cdot 4 H_2O$ .

Rh<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub> · 15 H<sub>2</sub>O

The yellow sulfate is produced on solution of moist hydrated rhodium (III) oxide (rhodium hydroxide) in dilute (1:10) sulfuric acid at temperatures not exceeding 50 °C. Then the hydrated Rh (III) oxide is precipitated from the cold solution with KOH (avoiding an excess of the latter) and washed on a membrane filter until the colloidal hydroxide passes through. Suction is then applied and as much water as possible is removed; the residue is dissolved without heating in dilute sulfuric acid.

The solid salt is obtained by evaporating this solution in vacuum, dissolving the residue in absolute alcohol, and precipitating with 10-20 times its volume of ether; this gives a pale-yellow, fine, flocculent residue. After filtration, washing and drying, this becomes a light, yellowish-white powder. The yield is always poor, at most 20%.

 $Rh_2(SO_4)_3 \cdot 4H_2O$ 

Red, amorphous rhodium sulfate is obtained either by evaporating a solution of the isomeric yellow salt or by precipitating hydrated Rh (III) oxide from a RhCl<sub>3</sub> solution at the boiling point, washing the precipitate with hot water, dissolving it in hot dilute sulfuric acid, and evaporating the mixture. To remove the excess  $H_2SO_4$ , the product is dissolved in  $H_2O$  and  $Ba(OH)_2$  is added until Rh (III) hydroxide begins to precipitate. The solution is filtered and again evaporated. The two sulfates undergo quite different precipitation reactions. Barium chloride precipitates the  $SO_4^{2-}$  almost quantitatively from solutions of the yellow salt prepared in the cold. These are acidic and KOH precipitates the rhodium from such solutions. On the other hand, the red salt solutions prepared under the same conditions either fail to give these reactions, or react only gradually, but in any case, not quantitatively. We must therefore conclude that in this last case we are dealing with a complex in which the bonds are stronger than in the yellow salt.

REFERENCE:

F. Krauss and H. Umbach. Z. anorg. allg. Chem. 180, 42 (1929).

# Chloropentaamminerhodium Salts [RhCl(NH<sub>3</sub>)<sub>5</sub>]Cl<sub>2</sub>, [RhCl(NH<sub>3</sub>)<sub>5</sub>](NO<sub>3</sub>)<sub>2</sub>

The preparation of the chloride and nitrate of these compounds, which are also known as purpureo salts, is given in the section on the preparation of pure rhodium (p. 1585 ff.).

#### Pure Iridium

Ir

Chemically pure iridium is best prepared by ignition of  $(NH_4)_2IrCl_6$ . To obtain especially pure material, the metal should be reconverted to  $(NH_4)_2IrCl_6$  (see p. 1594), which is then reignited.

PROPERTIES:

Very hard, fairly brittle metal; m.p.  $2454^{\circ}$ C. On ignition in air, forms small quantities of a volatile unstable oxide, IrO<sub>3</sub>; thus, under conditions of oxidizing ignition the weight of Pt-Ir alloys does not remain constant. Extraordinarily resistant to acids; insoluble even in aqua regia. Attacked with comparative ease by Cl<sub>2</sub>, particularly in the presence of NaCl, with which the nascent chloride forms a double salt.

## Iridium (IV) Oxide

# IrO2

In the method of Wöhler and Streicher,  $IrO_2$  is prepared from green  $IrCl_3$  which can be readily oxidized in a stream of  $O_2$  at 600°C, giving blue-black  $IrO_2$ .

The oxidation of fine iridium powder in a stream of air or oxygen does not give  $IrO_2$  quantitatively.

PROPERTIES:

Black to blue-black powder, insoluble in acids. Crystal structure: rutile type.

**REFERENCE:** 

L. Wöhler and S. Streicher. Ber. dtsch. chem. Ges. <u>46</u>, 1721 (1913).

# Hydrated Iridium (IV) Oxide

#### $IrO_2 \cdot 2 H_2O$

I. An aqueous solution of  $IrCl_4$  or  $H_2IrCl_6$ , prepared by the old method of Vauquelin (see  $H_2IrCl_6$ , method I, p. 1593), is evaporated several times to a sirupy consistency in a vacuum at 40 °C. After each evaporation, the sirup is redissolved in water; this treatment completely removes the excess of HCl. The final concentrate is again diluted, and dilute aqueous KOH is added in drops to the boiling solution until the color changes from dark red-brown to green and then to blue. The solution is then held at the b.p. for some time to oxidize any Ir (III) which may be present and to complete the precipitation of the hydrated Ir (IV) oxide. The residue of deep-blue coarse floc is filtered off, washed with water, then with absolute alcohol, and dried in a vacuum desiccator.

II. Gerlach's method consists in adding the KOH solution in drops to a boiling solution of  $Na_2IrCl_6$ , to give the alkali-free hydrated oxide. The use of excess hydroxide leads to a product which contains alkali. Purification is the same as in method I.

PROPERTIES:

Very dark-blue powder. The hydrated oxide prepared by method I may be converted to  $IrO_2$  at 350°C in a stream of  $N_2$ . Freshly precipitated  $IrO_2 \cdot 2$  H<sub>2</sub>O is soluble in acids.

REFERENCES:

- I. F. Krauss and H. Gerlach. Z. anorg. allg. Chem. 143, 126 (1925).
- II. H. Gerlach. Thesis, Tech. Hochschule, Braunschweig, 1925, pp. 4, 42; N. K. Pschenizyn and S. E. Krassikow, abstract in Chem. Zentr. <u>1933</u>, I, 3911.

# Hydrated Iridium (III) Oxide

#### Ir<sub>2</sub>O<sub>3</sub> · x H<sub>2</sub>O

A solution of  $Na_3IrCl_6$  is prepared either in the same way as in method II for  $K_3IrCl_6 \cdot 3 H_2O$  (see p. 1594) or, better, as suggested by Ogawa, from a solution of  $Na_2IrCl_6$  and sodium oxalate at 50°C according to the equation

 $2 \operatorname{Na}_{2}[\operatorname{IrCl}_{6}] + \operatorname{Na}_{2}C_{2}O_{4} = 2 \operatorname{Na}_{3}[\operatorname{IrCl}_{6}] + 2 \operatorname{CO}_{2}$ 

In either case the solution is treated with potassium hydroxide or potassium carbonate solution in a stream of  $CO_2$ . The separation, washing and drying of the hydrated oxide must be carried out under an inert gas ( $CO_2$  or  $N_2$ ). The alkali cannot be completely removed from the product.

PROPERTIES:

Pale-green to dark powder, depending on the precipitation conditions and water content; oxidized in air to the hydrated Ir (IV) oxide, particularly when damp. The Ir (III) compounds are more stable in acid solutions than the Ir (IV) salts; the reverse is true in alkaline solutions.

**REFERENCES:** 

L. Wöhler and W. Witzmann. Z. anorg. Chem. <u>57</u>, 334 (1908); Preparation of Na<sub>3</sub>IrCl<sub>6</sub> solution: E. Ogawa. J. Chem. Soc. Japan <u>50</u>, 246 (1929).

# Iridium (III) Chloride

## IrCl<sub>3</sub>

I.

 $Ir + l^{1/2}Cl_{2} = IrCl_{3}$ 192.2 106.4 298.6

Fine iridium powder is placed in a porcelain boat set in an open-end glass combustion tube. The gas inlet side of the tube is drawn to a small-diameter tubing, while the other end carries a ground-glass joint. An  $O_2$ -free stream of chlorine, containing a small percentage of CO, is passed through the tube, which is heated to about 600°C with a burner and illuminated either with direct sunlight or light from a burning magnesium ribbon. The chlorination is complete in about 15 minutes.

II. Alternatively,  $IrO_2 \cdot 2 H_2O$  is heated to 240°C in a stream of  $Cl_2$  and illuminated with sunlight or a burning magnesium ribbon.

III. Finally,  $(NH_4)_2IrCl_6$  may be decomposed in a stream of  $Cl_2$  at 440-550°C; the conversion of 0.5 g. requires two hours.

PROPERTIES:

Dark olive-green powder. Stable up to 760 °C under a Cl<sub>2</sub> pressure of 1 atm. (Streicher); at 700 °C the color changes to bright yellow.

**REFERENCES:** 

F. Krauss and H. Gerlach. Z. anorg. allg. Chem. <u>147</u>, 265 (1925); L. Wöhler and S. Streicher. Ber. dtsch. chem. Ges. <u>46</u>, 1720, 1582 (1913); S. Streicher. Thesis, Univ. of Darmstadt, 1913.

# Hexachloroiridic (IV) Acid

#### H<sub>2</sub>IrCl<sub>6</sub>

I. A solution of  $(NH_4)_2IrCl_8$  is decomposed by bubbling  $Cl_2$  through it at about 4°C; then the liquid is concentrated at 40°C (12-15 mm.) until a dark-brown sirupy mass results. This is allowed to stand for some time in an evacuated desiccator containing CaO (until it congeals and crystallizes). The low temperatures mentioned must be maintained to avoid the formation of NCl<sub>3</sub>.

II. A solution of  $(NH_4)_2IrCl_6$  is heated with aqua regia on a water bath (approximately 10 hours) until the  $NH_4^+$  is completely split off; the solution is repeatedly concentrated with conc. HCl until the HNO<sub>3</sub> is completely removed.

REFERENCES:

- I. Vauquelin. Liebigs Ann. <u>89</u>, 150, 225 (1845); A. Gutbier and F. Lindner. Z. phys. Chem. <u>69</u>, 304 (1909).
- II. S. C. Woo and D. M. Yost, J. Amer. Chem. Soc. <u>53</u>, 884 (1931).

## Potassium Hexachloroiridate (IV)

#### K<sub>2</sub>IrCl<sub>6</sub>

I. A mixture of fine iridium powder and twice its weight of KCl is heated in a porcelain boat almost to red heat while chlorine is

passed over it. After cooling, the excess KCl is extracted by washing with the least possible quantity of cold water. Then the double salt is dissolved in boiling water and filtered free of unconverted Ir. The solution is slowly evaporated in a porcelain dish. The  $K_2IrCl_6$  crystallizes as small, shiny, red-black octahedra, which yield a red powder on grinding.

In the above extraction with boiling water it is best to add a few drops of nitric acid in order to prevent the formation of  $K_3IrCl_8$  and convert any Ir (III) present to Ir (IV).

II. According to Puche, better results are obtained by allowing  $Na_2IrCl_6$  solution to react with solid KCl while a stream of  $Cl_2$  is bubbled through the mixture. The crystalline deposit is filtered off with suction and washed several times with dilute alcohol. It is then rapidly washed with some water. The product is dried in a drying oven at about 100°C.

PROPERTIES:

Deep dark-red octahedra. Solubility (20°C) 1.12 g./100 g.  $H_2O$ . Insoluble in alcohol. d 3.5.

**REFERENCES:** 

I. Old process of Berzelius: G. Gire. Ann. Chim. <u>4</u>, 210 (1925).

II. F. Puche. Ibid. 9, 270 (1938).

# Ammonium Hexachloroiridate (IV)

# (NH<sub>4</sub>)<sub>2</sub>IrCl<sub>6</sub>

A mixture of iridium metal powder plus twice its weight of NaCl is converted to  $Na_2IrCl_6$  by heating to 400 °C in a stream of  $Cl_2$ (compare the preparation of the analogous  $K_2IrCl_6$ ); this salt is dissolved in some water. Addition of  $NH_4$  Cl to this solution [or to other solutions of Ir (IV)] leads to the formation of  $(NH_4)_2IrCl_6$ ; the latter is only slightly soluble.

SYNONYM:

Ammonium iridium (IV) chloride.

PROPERTIES:

Dark-red octahedra. Solubility (cold) about 5 g., (100 °C) about 10 g./100 ml.  $H_2O.$  d 3.03. Crystal structure:  $K_2PtCl_6$  type.

REFERENCE:

A. Gutbier. Z. phys. Chem. 69, 307 (1909).

## Potassium Hexachloroiridate (III)

#### K<sub>3</sub>IrCl<sub>6</sub> · 3 H<sub>2</sub>O

I. A hydrogen stream is passed over gently heated (not over  $150^{\circ}$ C)  $K_2$ IrCl<sub>6</sub> placed in a quartz or porcelain boat; the reduction proceeds according to the equation

 $3 K_2 IrCl_6 + 3 H_2 = 2 K_3 IrCl_6 + Ir + 6 HCl$ 

II. A solution of  $K_2IrCl_6$  (the concentration should be as high as possible) in freshly prepared  $H_2S$  water is heated until the color turns olive-green. Then KCl is added and crystals of  $K_3IrCl_6 \cdot 3 H_2O$  deposit out; these can be dehydrated, if need be. The  $K_2IrCl_6$  can be reduced with  $SO_2$  in the same manner, but the product must be neutralized with  $K_2CO_3$ .

Alternate method: Reduction with oxalate: see section on  $Ir_2O_3 \cdot xH_2O$  (p. 1592).

SYNONYM:

Potassium iridium (III) chloride.

PROPERTIES:

Dark olive-green crystals. Readily soluble in water, insoluble in alcohol.

**REFERENCE:** 

I. F. Puche. Ann. Chim. 9, 273 (1938).

## Pure Ruthenium

#### Ru

Gutbier and Trenkner give the following method for the preparation of the pure metal. To start with, 30 g. of the fine metal powder is heated to dull red heat for three hours in a stream of  $O_2$  in order to volatilize Os, which is often present as an impurity. The partially oxidized Ru is then reduced in a stream of  $H_2$ ; then a mixture of the metal, the purest KOH, and the purest KNO<sub>3</sub> (3:25:3 by weight) is prepared and fused in a flat silver dish. The green melt is kept in the liquid state for half an hour. It is then cooled, and the reaction product is broken into small lumps and dissolved in lukewarm water. The orange-yellow solution is poured into a large retort, the neck of which is joined to a 2-m.long glass tube in such a way that the neck of the retort projects as far as possible into the tube. That tube is placed in a metal trough of about the same length filled with an ice-salt mixture. The other end of the tube is connected to a flask half-filled with 30% KOH. Then a fast stream of dry Cl<sub>2</sub> is introduced through the filler tube of the retort. So much heat of reaction is evolved that the RuO<sub>4</sub> distills over in a very short time. It is in the form of golden-yellow drops which solidify as a yellowish-red mass in the cooled condenser tube.

As soon as the formation of the tetroxide subsides, the contents of the retort are heated to 80-90 °C with a microburner while continuing to introduce  $Cl_2$ ; the whole operation is stopped only when a yellow vapor (a mixture of  $RuO_4$  and  $Cl_2$ ) can be seen in the attached flask. Since only ruthenium is capable of forming volatile compounds under these conditions (the osmium having been removed previously), all the other impurities remain in the retort.

To convert the tetroxide to the metal, the  $RuO_4$  is washed out from the tube with lukewarm water, transferred to a porcelain dish and, when completely dissolved, reduced immediately with pure alcohol (if the alcohol is added before solution is complete a violent explosion may occur!). The resulting inky liquid is concentrated on a water bath and the residue reduced to elemental Ru with pure  $H_2$ .

Alternate method: A method which is related to the analytical procedure of Wichers et al. separates the osmium by distillation of a nitric acid solution of the products of the alkali fusion step. The residue from this distillation is made alkaline and ruthenium is then distilled off, using  $Cl_2$  as above. Regarding the danger of explosions, see the properties of  $RuO_4$ .

PROPERTIES:

M.p.  $\sim 2400$  °C; d 12.43. Very hard and brittle; can be pulverized. When melted, part of the metal oxidizes and volatilizes as RuO<sub>4</sub>, which is stable at very high temperatures and gives off a peculiar choking odor.

REFERENCES:

A. Gutbier and C. Trenkner. Z. anorg. Chem. <u>54</u>, 167 (1905); E. Wichers, R. Gilchrist and W. H. Swanger. Trans. Amer. Inst. Mining Metallurg. Eng. <u>76</u>, 626 (1928).

## Ruthenium (IV) Hydroxychloride

#### Ru(OH)Cl<sub>3</sub>

Heating of  $RuO_4$  with conc. HCl on a water bath gives a darkbrown solution; according to Remy and Wagner, this contains the Ru (IV) chloride. If inhibited, the reaction (which involves the splitting off of  $Cl_2$ ) can be started by addition of a few drops of alcohol. The product solution is evaporated and a dark-brown product [Ru(OH)Cl<sub>3</sub>] is obtained. On the basis of an early incorrect assumption, this is often called "water-soluble ruthenium trichloride."

PROPERTIES:

Dark-brown salt. Very readily soluble in water. A certain proportion is apparently always present as RuCl<sub>3</sub>.

**REFERENCES:** 

A. Gutbier and C. Trenkner. Z. anorg. Chem. <u>45</u>, 167 (1905).
 H. Remy and A. Lührs. Ber. dtsch. chem. Ges. <u>61</u>, 917 (1928);
 <u>62</u>, 201 (1929);
 H. Remy. Ibid. <u>61</u>, 2110 (1928).

Ruthenium (III) Chloride

RuCl<sub>3</sub>, RuCl<sub>3</sub> · H<sub>2</sub>O

RuCl<sub>3</sub>

I. In the method of Remy et al., a mixture of carefully predried  $Cl_2$  and CO (initial ratio of 1:4) is passed over a boat containing ruthenium powder and placed in a Vycor combustion tube. After displacing all the air from the tube, the latter is heated to 700-800°C, and the fraction of  $Cl_2$  in the gas stream is simultaneously increased. The beginning of the reaction is clearly marked by a considerable swelling of the material. At the end of the reaction the boat is kept for half an hour at bright red heat, and the CO throughput is gradually reduced and finally stopped. Cooling in the  $Cl_2$  stream yields a well-crystallized product.

II. In the method of Wöhler and Balz,  $RuCl_3$  is prepared without the use of CO. A mixture of the metal and NaCl is heated at 700°C in a stream of  $Cl_2$ , after which the products are reduced with  $H_2$  at 400°C and extracted with  $H_2O$ . The finely divided, velvety-black metal obtained in this way is chlorinated at 800°C.

PROPERTIES:

Formula weight 207.5. Method I gives good crystals in the form of shiny black platelets. Insoluble in water.

REFERENCES:

- I. H. Remy (with M. Köhn). Z. anorg. allg. Chem. <u>137</u>, 365 (1924);
  H. Remy and Th. Wagner. Ibid. 168, 2 (1928).
- II. L. Wöhler and P. Balz. Ibid. 139, 413 (1924).

RuCl<sub>3</sub> · H<sub>2</sub>O

Since warm hydrochloric acid solutions of Ru (III) are partly oxidized by atmospheric oxygen (for example, on concentration) commercial "water-soluble ruthenium trichloride" is not free of Ru (IV). A pure product corresponding to the formula  $RuCl_3 \cdot H_2O$  can be obtained from this material by electrolytic reduction.

A cathode of platinized Pt foil  $(40 \times 35 \text{ mm.})$  and an anode of polished Pt foil are suspended inside a small porous clay cylinder placed inside a rectangular, 200-ml. glass trough. The cathode liquid is a 0.03 M solution of commercial RuCl<sub>3</sub> [or a solution of an evaporation residue which corresponds approximately to Ru(OH)Cl<sub>3</sub>] which is 2N in HCl; the anode liquid is 2N HCl. Efficient stirring is necessary. The electrolysis is carried out at 0.03-0.1 amp., with separate control of the cathode potential. The initial darkbrown color of the solution gradually clears. The electrolysis is stopped when the cathode potential becomes constant at 0.01 volt and H<sub>2</sub> evolution commences. The reduced solution should be red. A blue color indicates the formation of the undesirable Ru (II).

The RuCl<sub>3</sub> solution must be concentrated to crystallization in the absence of air, in order to prevent reoxidation by atmospheric oxygen. The evaporation is carried out in a round-bottom flask fitted with a dropping funnel; the flask is purged of air with HCl gas. The reduced solution is then introduced via the dropping funnel; with HCl continuously passing over it, it is evaporated at the boiling point to a sirupy thickness, and finally to dryness at 80-100 °C. It is finally dried to constant weight in a vacuum desiccator over H<sub>2</sub>SO<sub>4</sub>.

The salt obtained in this way is free of Ru (IV) and has the composition indicated by the formula.

REFERENCE:

G. Grube and G. Fromm. Z. Elektrochem. 46,661 (1940).

#### Ammonium Hexachlororuthenate (IV)

#### (NH<sub>4</sub>)<sub>2</sub>RuCl<sub>6</sub>

A concentrated solution of  $NH_4$  Cl is added to a ruthenium chloride solution and the resultant mixture concentrated in air. The dark-red crystalline powder is not homogeneous: it contains  $(NH_4)_2[Ru(OH)Cl_5]$ .

SYNONYM:

Ammonium ruthenium (IV) chloride.

## Ruthenium (IV) Oxide

#### RuO<sub>2</sub>

In the method of Remy and Köhn,  $\text{RuO}_2$  is prepared by heating fine ruthenium powder at about 1000°C in a stream of carefully predried  $O_2$ . It can also be prepared by ignition of  $\text{RuS}_2$  in air (the  $\text{RuS}_2$  is obtained by precipitation of ruthenium chloride solutions with  $H_2S$ ). Wöhler et al. suggest heating pure  $\text{RuCl}_3$  at 600-700°C in a stream of  $O_2$ .

PROPERTIES:

Dark-gray powder with a metallic luster, iridesces green and blue. Insoluble in acids. Readily reduced by  $H_2$  even at moderate heating. d 7.0. Crystal structure: rutile type.

REFERENCES:

H. Remy (with M. Köhn). Z. anorg. allg. Chem. <u>137</u>, 381 (1924);
 L. Wöhler, P. Balz and L. Metz. Ibid. <u>139</u>, 213 (1924).

## Ruthenium (VIII) Oxide

#### RuO4

Chlorine is passed through a solution of an alkali ruthenate, as described earlier in the section on the purification of ruthenium.

In another method (Ruff and Vidic) mixtures of ruthenium powder with KMnO<sub>4</sub> and KOH are fused; the ruthenate produced in this way is decomposed with  $H_2SO_4$  while still hot; a CO<sub>2</sub> stream is simultaneously passed through the reaction vessel and the RuO<sub>4</sub> distills off. A mixture of Ru, KMnO<sub>4</sub> and KOH (1:2:20 by weight) is fused to a mobile liquid. The dark-green melt is kept liquid for 0.5-1 hour after all the permanganate has been added. After cooling, the melt is dissolved in water and placed, with one additional part of KMnO<sub>4</sub>, in a flask fitted with a dropping funnel and containing 1:3 H<sub>2</sub>SO<sub>4</sub>. An ice-cooled flask containing some water is used as the first receiver (no alcohol! see properties), followed by a flask containing some 7% NaOH. Sulfuric acid is introduced until the color changes from green to red; then further H<sub>2</sub>SO<sub>4</sub> (1/3 of the total liquid volume) is added. After this a fast air stream is bubbled through while the solution is heated to 40-50 °C. Long golden-yellow needles of RuO<sub>4</sub> soon form in the ice-cooled flask. Later, ruthenate is formed in the NaOH solution, producing an orange-red color. Finally the solution is heated to boiling in order to steam-distill any remaining RuO<sub>4</sub>. The yield is almost quantitative.

PROPERTIES:

Formula weight 497.1. A solid composed of golden-yellow rhombic prisms. Very volatile, subliming even at room temperature. Characteristic odor comparable to nitric oxide or ozone; very irritating to the respiratory tract. Less irritating to the eyes than  $OsO_4$ . Melts at 25 °C to an orange-red liquid. Solubility (20 °C): 20.3 g./liter of H<sub>2</sub>O. Vapors and concentrated solutions tend to react explosively with organic substances, such as alcohol, filter fibers, etc. Distillation must therefore be carried out in perfectly clean equipment.

**REFERENCE:** 

O. R. Ruff and E. Vidic. Z. anorg. allg. Chem. 136, 49 (1924).

## Potassium Ruthenate and Potassium Perruthenate

## K<sub>2</sub>RuO<sub>4</sub> · H<sub>2</sub>O, KRuO<sub>4</sub>

A dark-green melt is obtained by heating a mixture of Ru powder with KOH and adding KClO<sub>3</sub> or KNO<sub>3</sub>; this readily takes up water, giving an orange-red solution. On evaporation,  $K_2RuO_4 \cdot H_2O$ crystallizes in iridescent green prisms which appear red when spread in thin layers and viewed by transmitted light. If Cl<sub>2</sub> is introduced into the red solution, the latter becomes green due to the formation of perruthenate. Continued passage of Cl<sub>2</sub> yields RuO<sub>4</sub>. However, if the Cl<sub>2</sub> stream is shut off at the right moment, KRuO<sub>4</sub> is precipitated on cooling as small black tetragonal crystals. In contrast to the ruthenates, the perruthenates are not stable above 200°C.

REFERENCES:

A. Gutbier, F. Falco and H. Zwicker. Z. anorg. Chem. <u>22</u>, 490 (1909); F. Krauss. Z. anorg. allg. Chem. <u>132</u>, 306 (1924).

#### Pure Osmium

#### Os

Osmium powder is purified by fusion in an oxidizing alkali melt. Nitric acid liberates the volatile  $OSO_4$  from an aqueous solution of this melt; the  $OSO_4$  is distilled in a stream of air into a receiver containing aqueous NaOH and is absorbed. The osmium is then reprecipitated as  $OSS_2$  and filtered off. It is reduced to the metal in a stream of hydrogen.

The procedure is almost the same as in the purification of ruthenium. The fine metal powder is mixed with KOH and  $KNO_3$  and fused at red heat. After cooling, the melt is dissolved in water in a retort. Nitric acid is added until the solution becomes acidic, and the  $OsO_4$  liberated is carried in an air or oxygen stream to a receiver containing aqueous NaOH, in which it is absorbed. The solution is treated with  $H_2S$ , which precipitates the osmium quantitatively (as  $OsS_2$ ). The precipitate is filtered off and reduced in a stream of hydrogen.

Since the sulfur is difficult to extract from the metal after the hydrogen reduction, the distillation receiver can also be charged with aqueous KOH (instead of NaOH), the resulting osmic acid salt may be reduced to  $K_2OSO_4$  with alcohol, and the  $K_2OSO_4$  reduced to Os with  $H_2$ .

PROPERTIES:

M.p. > 3000°C. Very hard and brittle, readily pulverized. The powder always retains the characteristic odor of  $OsO_4$ , since traces of the latter are formed in air even at room temperatures. Heating in air leads to complete combustion to  $OsO_4$ .

# Osmium (IV) Chloride

OsCl<sub>4</sub>

 $Os + 2 Cl_2 = OsCl_4$ 190.2 141.8 332.1

Small quantities of Os (prepared, for instance, by reduction of  $OsO_2$  with  $H_2$ ) are heated to 650-700°C in a porcelain boat set in

a glass combustion tube, while a slow stream of very pure  $Cl_2$  is passed through. The tube is constricted beyond the boat and lagged for 20 cm. with asbestos to produce a zone of gradual temperature drop. About 2 hours are required for 0.2-0.5 g. of Os to react. The chloride precipitates in various forms (crusts to powders) and in various colors (black to red-brown). It deposits in the reactor tube at and beyond the construction. The tube is melt-sealed at the constriction, and the part of the tube containing the chloride is evacuated, a cooling trap being inserted before the pump. The material is sublimed in vacuum, using the same tube, and deposited in a further section of the tube.

PROPERTIES:

Black crust with a metallic luster, or red-brown powder. Insoluble in water and other solvents and in concentrated oxidizing acids. Slowly hydrolyzed by water. The above product does not correspond exactly to the composition given by the formula.

REFERENCE:

O. Ruff and F. Bornemann. Z. anorg. Chem. 65, 446 (1910).

# Sodium Hexachloroosmate (IV)

# $Na_2OsCl_6 \cdot 2 H_2O$

In the method of Gutbier and Maisch, fine osmium powder and NaCl are mixed in a 1:1 ratio and the mixture heated in a porcelain boat in a stream of  $Cl_2$  for half an hour, at which point the temperature should correspond to a dull red heat. The conversion to  $Na_2OsCl_6$  is almost complete. The sintered contents of the boat are dissolved in the minimum quantity of cold dilute hydrochloric acid. The unreacted metal is filtered off and the filtrate is saturated with HCl (careful cooling). Most of the excess NaCl is thus separated although part of the  $Na_2OsCl_6$  also precipitates out.

Gradually evaporation of the filtrate yields  $Na_2OsCl_6$  as beautiful crystals, which, however, obstinately retain traces of NaCl even after repeated crystallization from dilute hydrochloric acid.

**REFERENCE:** 

A. Gutbier and K. Maisch. Ber. dtsch. chem. Ges. <u>42</u>, 4239 (1909).

# Ammonium Hexachloroosmate (IV)

#### (NH<sub>4</sub>)<sub>2</sub>OsCl<sub>6</sub>

In the method of Gutbier and Maisch, this salt is precipitated by allowing a dilute alcoholic solution of  $\rm NH_4Cl$  to react with the stoichiometric quantity of  $\rm Na_2OsCl_6$  (also in alcoholic solution). The precipitate is a fine, dark-red powder. It crystallizes from dilute hydrochloric acid (or from a mother liquor consisting of the components) in beautiful, shiny black octahedra which are opaque under the microscope.

Gutbier claims that  $(NH_4)_2OsCl_6$  is also obtained by dissolving in HCl the sublimates from osmium fusion, concentrating the solution, and mixing the filtered liquid with  $NH_4$  Cl.

REFERENCE:

A. Gutbier and K. Maisch. Ber. dtsch. chem. Ges. <u>42</u>, 4239 (1909).

# Osmium (IV) Oxide

# OsO<sub>2</sub>

Osmium (VIII) oxide is reduced in the cold by a stream of  $H_2$ . If, however, the OsO<sub>4</sub> is heated in the  $H_2$  stream, the product is the metal. Osmium (IV) oxide can also be prepared by heating a fine powder consisting of a mixture of  $K_2OsCl_6$  and three times its amount of Na<sub>2</sub>CO<sub>3</sub>. The temperature should be lower than red heat; the cooled product is extracted with water which is slightly acidified with dilute hydrochloric acid. The product is perfectly pure OsO<sub>2</sub>.

Alternate method: Heating of Os to 600-610 °C in a nitrogen stream saturated with OsO<sub>4</sub> vapor. Unreacted OsO<sub>4</sub> is reclaimed from the nitrogen stream by cooling it to a low temperature [O. Ruff and H. Rathsburg, Ber. dtsch. chem. Ges. <u>50</u>, 495 (1917)].

PROPERTIES:

Formula weight 222.2. d 11.4. Black powder. Insoluble in water and acids. Forms  $OsO_4$  on heating in air; readily reduced to the metal by  $H_2$ . Crystal structure: rutile type.

#### Osmium (VIII) Oxide

OsO4

 $O_{s} + 2 O_{2} = O_{s}O_{4}$ 190.2 64.0 254.2

Pure  $OsO_4$  is best prepared by a dry method. Osmium powder is heated in a boat placed in a glass or quartz tube through which a

stream of dry oxygen is passed. The metal burns to  $OsO_4$ , which deposits beyond the heated zone of the tube or, better, in a bulb fused to the tube and cooled in ice. The deposit consists of white shiny crystals, though at first it may be a liquid (occasionally pale yellow in color), which forms a crystalline solid on cooling. Two or three receivers, preferably connected via ground-glass joints, are fitted to the glass tube beyond the bulb. They are halffilled with KOH to absorb the OsO<sub>4</sub> vapor entrained by the oxygen. The OsO<sub>4</sub> in the receivers is reclaimed by reduction to potassium osmate (violet-blue octahedra); this is accomplished by treating the combined caustic liquors from the receivers with an equal quantity of alcohol.

The temperature at the boat is increased gradually so that the reaction does not proceed too vigorously, heating initially to 300°C and gradually increasing the temperature to 800°C. The temperature is then slowly reduced, and the product allowed to cool in the tube. The heating is most conveniently carried out in a small tubular electric furnace.

PROPERTIES:

M.p. 40.6-40.7 °C, b.p. 130 °C; d 4.9. Soluble in water without decomposition; may be volatilized in steam. Dissolves slowly. Decomposed by conc. HCl with evolution of  $Cl_2$ . A solution of  $OsO_4$  is not decomposed by light and can be indefinitely stored in transparent bottles. Toxic; the vapor first irritates the respiratory passages and (particularly) the eyes. Decomposed in a stream of  $H_2$  at red heat, forming a mirror.

# Potassium Osmate (VI)

# $K_2OsO_4 \cdot 2 H_2O$

A solution of  $OsO_4$  in potassium hydroxide solution is reduced with alcohol.

Osmium powder (2 g.) is heated with 5 g. of KOH and 3 g. of KNO<sub>3</sub> in a silver dish to form a smoothly flowing melt. After cooling, the brown solid is dissolved in 50 ml. of water. The gray-violet crude salt is precipitated by adding twice the volume of alcohol. It is readily decomposed and cannot be recrystallized from water. It is decomposed by heating with 5 g. of CrO<sub>3</sub> and conc.  $H_2SO_4$ , the OsO<sub>4</sub> distilling off is collected in 10% KOH, and  $K_2OSO_4 \cdot 2 H_2O$  is precipitated from the resulting solution by adding an equal volume of alcohol. The solid is filtered off with suction, washed with 50% alcohol and with absolute alcohol, and dried in vacuum over  $H_2SO_4$ .

In a simpler method, osmium powder is heated directly in a stream of  $O_2$  (compare preparation of  $OsO_4$ ), and the  $OsO_4$  vapor is collected in 10% KOH.

PROPERTIES:

Pale violet-red octahedra. Readily soluble in water, insoluble in alcohol and ether. Stable only in dry air. The water of crystallization is removed by heating to  $200^{\circ}$ C in an inert gas. Heating in air produces OsO<sub>4</sub>.

**REFERENCE:** 

O. Ruff and F. Bornemann. Z. anorg. Chem. 65, 434 (1910).

# Potassium Osmiamate

#### K(OsO<sub>3</sub>N)

 $O_{5}O_{4} + KOH + NH_{3} = K(O_{5}O_{3}N) + 2 H_{2}O$ 254,2 56.1 17.0 291.3 36.0

In the method of Joly, 100 g. of  $OsO_4$  is dissolved in a solution of 100 g. of KOH in 50 ml. of  $H_2O$ ; the solution is heated to 40°C and dilute ammonia is added; this clears the dark-brown liquid and precipitates  $K(OsO_3N)$  as a granular, yellow crystalline powder.

Excess ammonia should be avoided because it may produce  $NH_4$  (OsO<sub>3</sub>N). The product is washed with some cold water and recrystallized. Larger crystals may be obtained by gradual evaporation of the solution; however, these crystals are dark due to incipient decomposition.

PROPERTIES:

Fine, granular, yellow crystals. Readily soluble in water, only slightly soluble in alcohol. Darkens on heating to 180°C, with decrepitation at higher temperatures. d 4.5. Tetragonal crystals.

**REFERENCE:** 

A. Joly. Comptes Rendus Hebd. Séances Acad. Sci. 112, 1442 (1891).

# Part III Special Compounds

# SECTION 1

Adsorbents and Catalysts

**R.WAGNER** 

# Introduction

Solid adsorbents or catalysts must possess large surface areas to allow contact with large quantities of reactants. Large surface areas can be obtained via two methods:

1) SUBDIVISION OF THE SOLID

The solid is subdivided into very small granules so that a large fraction of the total crystal lattice structure becomes exposed as particle surface. Comparison of x-ray and ultramicroscopic data then shows whether the resulting granules are primary particles (i.e., coherently diffracting single crystals) or secondary particles (i.e., a mosaic consisting of several primary crystals).

#### 2) CREATION OF A NETWORK OF INTERNAL PORES

In this case, the solid is permeated by a system of pores (interconnected or not), somewhat in the manner of a sponge. The net result is the creation of a large internal surface. The pore openings of such active solids should not be too narrow since they must allow the gases to penetrate into the interior (see [8]).

The two methods of achieving high surface may be illustrated on the classic catalyst, platinum; thus, platinum black is the subdivided solid, while platinum sponge is the porous form.

Active substances not only must have a large surface area, but must also possess a proper surface structure. As a general rule, one can expect the surface lattice of metals and ionic compounds to differ from the interior lattice of the crystal [12, 13, 14]. Thus, the active surface is very readily affected by external agents (such as impurities), and is also subject to other influences, such as the method of preparation, etc. In addition, it is often found that the same particle may carry several crystallographically differing surfaces. Obviously, these will differ not
only in their chemical properties [2, 4, 5] but also in their catalytic activity [7, 9, 11]. This fact, as well as the presence of intrinsic and impurity defects in the lattice, results in an overall surface which is usually very heterogenous; this, in turn, affects its adsorptive and catalytic behavior.

The methods of controlling particle size, surface area and surface structure during preparation of various adsorbents and catalysts are given in the preparative directions for individual substances, as well as in the general notes. A more extensive treatment of these problems is given in Ref. [A], especially in the articles in Vol. 4. The usual methods are frequently employed to obtain substances whose activity is not only related to their particle size and surface structure, but is also a direct consequence of other factors such as lattice defects, amorphism, the existence of unstable modifications [1, 10], etc.

We have seen in the above that the activated state of solids. which is the result of the existence of special conditions in the material, is rather unstable and can easily be destroyed. In the preparation of activated solids, this state is fixed by removing the conditions favorable to a transition to a more stable form (for example, slow aging at preparative conditions); i.e., the solid is "frozen" in the activated state. This may be done, for example, by rapid quenching, quick removal of supernatant mother liquors, etc. The active state also implies higher than normal surface energies. For this reason, active materials are generally very reactive, and are frequently used in heterogeneous reactions (solid-solid, solid-liquid, or solid-gas). They are more readily decomposed chemically than inactive preparations; thus, active metals oxidize faster, oxides hydrate more easily, hydroxides and hydrated oxides are more sensitive to CO<sub>2</sub>. All of these solids decrease in activity with time, due to a slow healing of surface defects and an eventual increase in grain size. The kinetics of such aging processes are in some ways analogous to those of the ion-hole processes in semiconductors. Aging proceeds via a series of individual steps and, depending on the activation energy of these steps, different optimum temperatures are required if the aging is to proceed at a significant rate. The temperature scale of Hüttig [6], derived as an extension of the work of Tammann (see table), is based on studies of metals and ionic compounds and provides a useful rule-of-thumb guide to the temperatures at which these processes take place.

It is seen from the table that some healing of surface defects is possible without undue reduction of particle size. In general, however, the temperature of any heat treatment of active materials must be strictly controlled to avoid deleterious effects. In heterogeneous gas catalysis, reaction temperatures exceeding those recommended in the table are often unavoidable; this leads to a

Period	α*	Processes occurring in the catalytic material		
Initial surface degradation	< 0.23	Reduction in adsorbing surface; degradation of those surface de- fects which possess the highest energy.		
Surface activation	0.23-0.36	Degradation of surface defects.		
Deactivation of the surface	0.33-0.45	Formation of a surface which is stable in a thermal equilibrium; beginning of particle sintering.		
Activation of the crystal center	0.37-0.53	Degradation of defects in the in- terior of the crystal.		
Deactivation of the interior of the crystal	0.48-0.8	Accretive crystallization.		
Relaxation and disintegration of the crystal	> 0.8	Stage prior to melting.		

 $\alpha = T/T_m$ , where T is the temperature of the experiment and  $T_m$  the melting point of the substance (°K).

rapid inactivation of the catalyst. In such cases a stabilization of the surface and of the remainder of the defect structure may be achieved by precipitating the catalyst onto a suitable carrier substance. This method is also used to transform into a quasi-solid form substances that, when pure, normally exist only in a subdivided form.

Such carriers must have good accessibility to gases, combined with reasonable mechanical strength and thermal stability. The frequently used carriers are:

#### NATURAL MATERIALS

Pumice, kieselgur, various silicates (asbestos, meerschaum, etc.), adsorbent clays, etc.

#### SYNTHETIC MATERIALS

Magnesium oxide,  $\gamma$ -aluminum oxide, synthetic rutile, thorium dioxide, silica gels, barium sulfate, activated carbons, metallic network supporting structures, various silicates (especially of Mg, Al), etc.

Natural materials are transformed into carriers in a variety of ways such as slurrying, washing and treatment with acids or alkalies; or they may have to be fractionated to separate the most active structures before the catalyst itself is deposited. Catalytic substances containing alumina, silica, thoria and similar carriers may also be obtained by coprecipitation.

It is frequently observed that the activity of a catalyst varies with the carrier and substrate and that certain catalyst-substrate combinations give especially good results (see [3]). This is a particular case of catalyst promotion which is frequently observed in mixed catalysts. This phenomenon is of great practical importance. It permits the creation of catalyst mixtures that are very active and capable of influencing reactions in a very specific manner, something that the individual components of the combination cannot achieve qualitatively or quantitatively.

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- [B] Advances in Catalysis, Edited by W. G. Frankenburg, V. I. Komarewsky and E. K. Rideal, New York, beginning with 1948.
- [C] Reviews of patents dealing with the preparation of adsorbents and catalysts appear from time to time in Kolloid-Zeitschrift.
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## ACTIVE METALS

The usual methods for the preparation of active metals fall into three groups, which are characterized by common preparative methodology and the same type of defect structure of the products.

#### PREPARATION BY REACTIONS OF SOLIDS

These reactions should be carried out topochemically; i.e., the active metal should be formed in the boundary region of the solid starting substances and not via a reaction between dissolved or gaseous particles. It is also desirable to avoid transport of the atoms of the solid from the initial reaction site; thus, any regrouping of atoms due to the reaction should involve minimum displacement. The desired product should be a loose network of mutually joined primary crystallites. The lower the temperature, the shorter the exposure of the material to high temperature; and the looser the structure of the starting material, the closer the approach to this ideal condition.

Reactions of this type include:

1) Reduction of solids with gaseous agents (see preparation of pyrophoric cobalt, p. 1615; Ni-Mg mixed oxalate catalyst, p. 1615).

2) Reduction of solids with solid reducing agents, solid-solid reactions (see tungsten, p. 1622).

3) Reduction of solids with solutions of reducing agents (see "molecular" silver, p. 1623).

4) Leaching out one component from a solid mixture (see Raney Ni, p. 1625).

5) Thermal decomposition of solids, resulting in liberation of a metal (see nickel formate-paraffin catalyst, p. 1631).

## PREPARATION OF ACTIVE METALS BY DEPOSITION FROM A HOMOGENEOUS MEDIUM

These reactions give materials with a broad particle size distribution, which does not follow a predetermined probability function, but is controlled by the processes of nucleation and phase formation. A precondition for such a distribution is a high degree of supersaturation of the homogeneous phase, something which is readily achieved given the poor solubility and high boiling of metals. One thus obtains many nuclei. Local supersaturation is insufficient and is to be avoided. Processes of this kind involve:

1) Reductions of compounds from the gaseous phase and from homogeneous solutions (see active copper, p. 1633).

2) Thermal decomposition of volatile metal compounds, especially carbonyls (see carbonyl iron, p. 1636).

### PREPARATION BY PRECIPITATION ON INTERFACES

The structure of such precipitates can be influenced by the carrier. Thus, oriented deposits are known in which the crystallites of the deposit are preferentially attached to a specific crystal plane of the carrier. Further, there exists the phenomenon of epitaxy, in which the crystal axes of the individual deposit particles have a definite spatial and geometric relationship to each other and to the crystal axes of the carrier. Under such conditions, there may occur significant changes in the relative proportions of some crystal surfaces to the total surface area. Precipitation on a surface is not necessarily preceded by a chemical reaction.

Among such methods of preparations are:

1) Electrolytic preparation of finely divided and active metals (see explosive antimony, p. 1638).

2) Electrochemical reduction, cementation (see silver, p. 1641).

3) Deposition from a vapor (see metallic deposits from a vapor, p. 1643).

Pure metallic preparations normally do not have a very high intrinsic activity; the total activity of a catalyst depends very much on the development of the surface. Lattice imperfections are usually observed only in the presence of impurities (incomplete reaction of starting materials) or in metals supported on carriers. The carriers prevent sintering of metal particles [73] on heat treatment during preparation or use, and in addition they stabilize crystal modifications beyond their normal range of existence.

The extremely active metals are pyrophoric; that is, they oxidize spontaneously on contact with air or in a high-temperature environment, becoming brightly incandescent (spontaneous and latent pyrophoric tendencies, respectively). The spontaneous pyrophoric tendency. which causes some obvious difficulties in the handling of these materials, may be converted to the latent one by mixing the products with a 0.5% solution of acetyl cellulose in acetone or a very dilute solution of polystyrene in benzene, followed by evaporation of the solvent. Frequently, a spontaneously pyrophoric metal may be sufficiently deactivated either by shaking it for some time with pure benzene, petroleum ether, ethanol or a similar substance, or by allowing such a liquid to evaporate from the mixture. On such treatment the particle surfaces become covered with a thin layer of oxide due to slow diffusion of oxygen through the liquid or due to slow exposure on evaporation. Because the exposure is slow and the particle is at least partly submerged in a heat-removing liquid, the heat generated by the oxidation does not increase the temperature to the point of ignition.

## **Pyrophoric Cobalt**

#### I.

 $2 \operatorname{CoO(OH)} + 3 \operatorname{H}_{2} = 2 \operatorname{Co} + 4 \operatorname{H}_{2} \operatorname{O}$ 183.9 67.2 l. 117.9

Cobalt (III) hydroxide (prepared as on p. 1520) is placed in a porcelain boat and reduced in a stream of hydrogen. The boat is heated by a tubular electric furnace whose temperature is regulated by a thermocouple connected to an on-off relay. For practical purposes, a temperature exceeding 300°C gives a sufficiently high reduction rate. The crystal structure and pyrophoric nature of the product (at room temperature) are related to the reduction temperature as follows:

Temperature, °C	300	400	500	600	700	800
Crystal structure	$\alpha$ -Co (hexagonal [49]) $\beta$ -Co (cub					bic)
Pyrophoric tendency (see p. 1614)	spontaneous		latent n		oyrophor	ic

# II. CARRIER-SUPPORTED PYROPHORIC COBALT, BY REDUCTION OF A COPRECIPITATE

A solution of 75.0 g, of Al(NO<sub>3</sub>)<sub>3</sub>  $\cdot$  9 H<sub>2</sub>O (0.2 moles) in 300 ml. of water is prepared and 200 ml. of 20% sodium hydroxide solution is added with efficient stirring. The initial precipitate is redissolved and a solution of 29.1 g. of  $Co(NO_3)_2 \cdot 6 H_2O(0.1 \text{ moles})$ and 20 ml, of conc. HNO<sub>3</sub> (d 1.40) in 500 ml. of water is immediately poured in (thin stream, good agitation). The violet-rose precipitate is allowed to settle and then is washed 4 or 5 times by decanting with pure water. It is then centrifuged off and dried in an oven at 75°C. The coprecipitate is then ground under water and boiled several times with water (250 ml. each time) until the absence of nitrate in the product can be established by some qualitative test reaction. The product is again collected by centrifugation and dried at 75°C, then at 100°C. Reduction of such coprecipates by method I yields spontaneously pyrophoric materials even at the highest reaction temperatures. While  $\beta$ -Co prepared by method I is converted more or less completely to the  $\alpha$ -form on grinding in the absence of air in an agate mortar, materials prepared via method II remain completely unchanged [17] on such grinding. The  $\beta$ -form appears to be the more active hydrogenation catalyst [56].

## Ni-Mg Mixed Oxalate Catalyst (1:1)

 $(Ni, Mg)C_2O_4 \cdot 2H_2O \xrightarrow{(H_2)} Ni/MgC_2O_4$ 

In the Langenbeck method [34] a solution of 15 g, of Ni(NO<sub>3</sub>)<sub>2</sub>. 6 H<sub>2</sub>O and 70 g, of Mg(NO<sub>3</sub>)<sub>2</sub>. 6 H<sub>2</sub>O in 600 ml, of water is heated to  $50^{\circ}$ C and the mixed oxalate precipitated by addition of a solution of 29.6 g. of  $H_2C_2O_4 \cdot 2 H_2O$  in 400 ml. of water (constant stirring). The mixture is then left standing for 12 hours to complete the crystallization. The compound is collected by suction filtration and the light-green crystals washed with water until free of nitrates. Drying at 100°C yields 16 g. of the mixed oxalate.



Fig. 336. Preparation of mixed salt catalysts by fluidized bed decomposition of oxygen-containing compounds. a reactor, b fritted-glass plate (distributor), c electric heating coil, d thermometer connected to an on-off relay in the heater circuit.

The material is reductively decomposed in the apparatus of Fig. 336. The sample is placed over the fritted-glass plate b (which acts as a distributor). When the hydrogen stream is adjusted to 10 liters/hour at STP, the mixed oxalate should form a stable fluidized bed. The temperature is then raised to 350°C. The decomposition takes 150 minutes.

PROPERTIES:

Black, pyrophoric powder. Aside from the metallic Ni produced in the decomposition, contains a nearly unchanged magnesium oxalate carrier [32]. Extremely active hydrogenation catalyst.

#### GENERAL:

The reduction of oxygenated compounds to active metals presented above is a very general method. Hydrogen is frequently the only useful gaseous reducing agent. It possesses a high thermal conductivity, and therefore the heat of reaction tends to be removed as fast as it is generated. This is important in systems where the metal, once produced, catalyzes another further reaction [2, 52]. Apart

from this, fewer undesirable side reactions can be expected with hydrogen than with other (possibly) useful gases such as CO, the lower hydrocarbons and  $NH_3$ , with which formation of carbonyls and contamination with carbides and nitrides is possible [20]. The flow rate must be sufficiently high to remove the volatile decomposition products as fast as they are formed; otherwise the reaction may be inhibited and the activity of the final product may be less than the optimum [56, 69, 71]. If necessary, the exit gas composition may be monitored [2]; a continuous monitoring system based on the thermal conductivity of the gas can be especially useful [8].

The oxygenated starting material must also be carefully chosen. In the following, we shall present some remarks pertaining to individual classes of starting materials.

## OXIDES

The best starting materials are active oxides, possibly produced *in situ*, preferably from hydroxides, hydrated oxides or carbonates. In some cases it may be necessary to start with a very well-defined oxide modification in order to obtain an active catalyst [29]. Occasionally, the required starting oxides are produced by thermal decomposition of nitrates. However, the activity of such products is not very high and their maximum specific surface does not exceed a few m.<sup>2</sup>/g. For this reason, nitrate decomposition is important only for the production of supported catalysts (see below). Oxides calcined at a high temperature, as well as spineltype materials, should be avoided, since their reduction times tend to be extremely long [2].

## FORMATES AND OXALATES

Heavy-metal salts in this class may reduced directly since their anions also are reducing agents and thus promote the overall reaction [36]. In some cases (see p. 1665), simple thermal decomposition of the formate or oxalate will yield the metal; in such cases the hydrogen acts only as a protective gas which prevents reoxidation. Metals obtained by this method are not always completely free from carbon [33, 37]. The nature of the starting material may influence the activity of the product metal catalyst to some extent; this is especially true if the material is reduced at a temperature just sufficient to effect the reaction [42].

The following methods apply to the production of carriersupported catalysts:

a) **Precipitation** of the compound to be reduced (hydroxide and carbonate by precipitation from solution, oxide by nitrate decomposition) on the desired carrier. The major methods involved are those of Sabatier [60, 61] and that illustrated by the case of active copper [47] (see copper tower, p. 459). Many industrial catalysts are prepared in this manner; it is especially recommended for cases where the active material must participate in a stoichiometric reaction.

b) **Coprecipitation**. In this case, the noble and the base metal are both attached to the same type of anion. The noble metal is in the form of a compound which yields the actual catalyst upon reduction, while the base-metal compound yields the carrier upon

reductive decomposition. The coprecipitation must be carried out in such a way as to avoid possible fractionation of the two components [66]. Thus, it is desirable that the coprecipitants form compounds or solid solutions under the mother liquor. It is known for instance, that divalent metal couples (Co/Zn; Ni/Zn[12]) as well as mixed bi- and trivalent metal couples (Mn/A1; Co/A1 [13]; Ni/A1 [13, 43, 44, 46]; Cu/A1 [5]) can form double hydroxides. Solid solutions (or mixed crystals) tend to give especially finely divided active metal catalysts, because this subdivision tends to exist in the material even prior to the reduction. Systematic studies by Langenbeck have shown that mixed formate and oxalate crystals tend to give especially active catalysts. Such mixed salts decompose at low temperatures and the active metal exists in a finely divided form [56]. The specific surface in such cases is high (in isolated cases it may exceed 200 m.  $^2$ /g. [54, 56]).

To achieve homogeneity with these relatively soluble compounds, which, however, have different solubilities, the mixed formates must be prepared by a special spray-drying technique [31]. The mixed oxalates, which are precipitated with oxalic acid and ammonium oxalate rather than with alkali oxalates (this tends to give more active catalysts), yield, as a rule, homogeneous materials via a simple precipitation. Complex oxalates are just as usable as the mixed oxalates [33].

c) Activation of the surface of suitably shaped metals by surface oxidation and reduction. This process may have to be repeated; highest activation is usually obtained after 3-4 cycles [21]. The product particle consists of a small solid metal nucleus—the carrier—to which the reduced metal, in finely divided or porous form, adheres tightly. The roughening of the surface accompanying this process leads to a useful increase in specific surface. The starting materials are usually thin metal foils, but sometimes oxidized wire cuttings (see CuO "wire" [45]) give a useful catalyst with a carrier of exceptionally high thermal conductivity. This is a useful feature because it helps achieve a uniform temperature distribution within a closely packed catalyst mass.

The above methods may also be used for the production of alloy powders and solid metallic solutions. However, the mechanical properties [24], the tendency to lattice imperfections [25], and the catalytic activity [4, 68] of such preparations are not simple functions of the composition.

The hydrogen reduction method is also used for the preparation of active forms of lower oxides and sulfides of multivalent metals. However, these materials require a much longer time to achieve reduction, even though the procedure is otherwise identical.

The optimum reduction temperature depends on a variety of factors. Among these the nature of the metal is, of course, of primary importance. However, the type of the anion, the purity of the starting material [51], the degree of decomposition and the defect state of the compound to be reduced are also important determinants of the optimum temperature. The deposition on a carrier or the presence of admixtures [11, 55, 62] may decisively influence the reductive behavior. The lowest reduction temperatures are achieved with specially purified gases (carefully dried  $H_2$  [56], CO free from CO<sub>2</sub>), if necessary, at reduced pressure [48].

Metal	or	Starting material for the reduction						
alloy	alloy		Nitrate	Hydroxide	Car- bonate	Formate	Oxalate	
Fe	, II	[28]	_	[18] a, b [29]		[3]	[3,6,37]	
Fe—Cr Fe—Mn	} I	[28]	-		-			
Fe—Co Fe—Ni	Í	Ξ	[53]	_	[68]	[3, 38, 40] [38]	[25]	
Fe—Cu	I			[57]			[24]	
Fe—Ag Fe—Au Fe—Pb	} I	-	-	[57]				
Co Co—Ni	I II I	=	[53,59] a [59] [53]	[17] b [17] —	a [27]	[39,41,42,56] b [56] [38,40]	[39, 42] b [34] —	
Ni	I II	[41] c [70]	[15,42,53] a [64,65]	[19,41,42] b [14,50,63]	[4, 19] a [7]	[39,41,42] a [58] b [30,31,35, 54]	[9, 39, 41] b [32, 34, 35]	
NiCu	I		[53]		[4]	-		
Cu	I II	[16] c [1,21]	a [26,67]	a [16,47]	[16]	b [33]	[72] b [33]	
W	I	[22,23]				_		

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- I: free metal or free alloy.
- II: metal on supported carriers: a deposited by precipitation; b deposited by coprecipitation; c obtained by activation (see text).

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#### Tungsten

## $WO_3 + 3Zn = W + 3ZnO_{231.9} + 196.1 + 183.9 + 244.1$

Small amounts of a homogeneous mixture of 50 g. of finely powdered calcined WO<sub>3</sub> (prepared from analytical grade sodium tungstate) and 150 g. of fine zinc dust (containing as little oxide as possible and dried at 150°C) are pressed into unglazed porcelain crucibles. An unglazed porcelain tube may be used when a larger batch is to be prepared; the tube, centered vertically in the crucible, provides better heat conduction. The mixture is then covered with a 1- to 2-cm, layer of zinc dust, and the crucible is then closed with a closely fitting asbestos lid. To initiate the reaction, the crucible is heated to 500-520°C in an electric furnace. As soon as the mixture ignites and a bright glow is visible through a small hole in the asbestos cover, the current is cut off. The reaction is completed within a few minutes and the crucible may then be removed from the furnace. After complete cooling, the crucible is broken up and the product added (in small portions) to cold, dilute (1:4) hydrochloric acid. The mixture is boiled until hydrogen evolution ceases and the supernatant is of Zn-free acid must be added from time to time). The product is washed in a centrifuge with oxygen-free water; at the end, the wash liquors must be free of chloride ion. Toward the end of the washing procedure, the metal begins to form a colloidal suspension. It should be covered with water at all times to avoid reoxidation. After washing, the water is displaced with ethanol, under which the

product may be stored in active form; or it is inactivated with benzene (see p. 1614) and dried in vacuum over  $P_2O_5$ . In the latter case, the product consists of coarse lumps, which may be cautiously triturated under benzene to give a fine powder upon drying [1, 3, 5].

PROPERTIES:

High density black powder which consists of 99% W, provided the workup has been rapid and no oxygen contacted the product. The average size of the primary particles is about 400 A, with the individual crystallites showing a slight lattice distortion [3]. Preparations which are not inactivated prior to storage oxidize in air, evolving heat; after this, they show oxygen bands in the powder pattern.

GENERAL:

This method is obviously applicable only to very high-melting metals which retain their subdivision and defect state in spite of the high reaction temperature. The retention of these properties is aided by the fact that the particles of the reaction product are embedded in ZnO, which, together with the unreacted Zn, gives effective protection against penetration of atmospheric oxygen during cooling. This form of tungsten, as well as the analogously prepared molybdenum [4], consists of such small particles that it gives colloidal solutions on peptization by the etching method (alternate treatment with dilute acids and bases).

Other solid reducing agents include metal hydrides (for example,  $CaH_2$ ) [2] and carbon. However, reduction with carbon does not yield solid oxidation products and introduces the danger of carbide formation.

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''Molecular'' Silver

In the method of Gomberg and Cohne [3], pure, thoroughly washed silver chloride is placed in a beaker and covered with water. A platinum disk, attached to a thoroughly flame-cleaned platinum wire, is embedded in the AgCl. A porous clay cell, closed at the bottom and containing some water and a few Zn rods, is placed on top of the AgCl. The reaction starts as soon as the protruding platinum wire is connected to the zinc rods. To increase the reaction rate, a few drops of HCl are added to the clay cell. Reduction of 250 g. of AgCl requires a few days. To decrease the migration of impurities from the zinc into the silver, the liquid level in the clay cell is always kept below that in the beaker. After completion of the reaction, the product (a metal-containing sludge) is washed with water, ammonium hydroxide, again with a large quantity of water, alcohol and finally ether.

PROPERTIES:

High-density gray powder. The individual particles are permeated with many pores (pore radius of the order of  $10^4$  A); the interconnected single grains are primary particles [8].

GENERAL:

The reduction of a suspended solid is applicable only to noble metals, but under favorable reaction conditions it produces highly dispersed materials [2]. Here again it pays to use very active starting materials, preferably prepared (by precipitation) immediately prior to use. Apart from the galvanic reduction method, one can use dissolved reducing agents. However, these must be absorbed to some extent by the precipitate to be reduced. If the precipitate is unable to absorb the reducing agent, the ions are reduced in solution, with consequent loss of the topochemical nature of the reaction [5].

Reactions of the above type give Cu from hydrated Cu oxide and  $N_2H_4$  [2]; Ag from Ag<sub>2</sub>O and  $H_2O_2$  [7]; Ag from AgCl and  $H_2CO$  [9],  $NH_2OH$  [5],  $N_2H_4$  [6],  $Cr^{2+}$  [4]; Pt from PtO<sub>2</sub> and  $H_2$ [1]. Platinum-asbestos and palladium-asbestos may also be prepared by this method (see p. 1563).

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#### Raney Nickel

### I. METHOD OF PAUL AND HILLY [29]

A fireclay crucible is charged with 400 g. of Al. It is then heated to 1200°C, and 300 g. of nickel cubes are added at once to the Al melt. Nickel cubes are especially useful in this case since the material is porous and thus quickly dissolved at the initial, comparatively low temperature. The nickel dissolves in a vigorous reaction which raises the temperature of the melt to 1500°C. The alloy must be prepared under a salt melt layer or in an inert atmosphere to protect the Al from oxidation.

After cooling, the alloy is broken up or cut on a lathe (however, it is best to grind it in a ball mill). Then, 250 g. of the powder is added in small portions to one liter of ice-cold, 25% aqueous NaOH. During the initial vigorous reaction the flask is kept in ice; then, the mixture is gradually heated to  $90-100^{\circ}$ C and this temperature is maintained until hydrogen evolution ceases. The solid is allowed to settle, the spent hydroxide solution is removed, and the process is repeated twice, each time with one liter of fresh base. The Ni sludge is then washed by decantation with water until the wash water is neutral to phenolphthalein. At the end, the water is displaced with ethanol or dioxane.

#### II. RANEY NICKEL W-6 [2]

The reactor is a two-liter Erlenmeyer flask equipped with a thermometer and a stainless steel stirrer. This flask is charged with 160 g. of NaOH and 600 ml. of water. The solid is dissolved with intensive stirring and the solution is cooled in an ice bath to 50 °C. Then, 150 g. of Raney nickel-aluminum alloy (1:1) is added in small pieces. The rate of addition should be such that the temperature of the mixture remains constant at  $50 \pm 2$  °C. The addition takes 20-30 minutes. The solution is then stirred for an additional 50 minutes while the temperature is kept at 50 °C (first by cooling and later by heating on a water bath). The catalyst sludge product is washed three times by decantation with water. It is immediately placed in the washing tube c of the apparatus in Fig. 337 (the last of the product is transferred into c with a stream of water from a wash bottle). Tube c and the one-neck Woulfe flask b are filled with water and the apparatus is assembled as quickly as possible. All rubber stoppers and tubing should be held in place with clamps or wires. Then,  $O_{z}$ -free hydrogen is introduced via e until the entire apparatus is under a gage pressure of 0.5 atm. This pressure is then maintained while stirring at such a rate that the catalyst is fluidized to a height of 18-20 cm. above the bottom of tube c and the wash water flow rate from b is 250 ml./min. When the water reservoir b is nearly empty, stopcocks g and a (the latter is connected to a large pressurized water reservoir) are simultaneously opened, and b is replenished at the same rate as water runs out at g (the flow rate is checked by a differential manometer).

In this way, 15 liters of  $H_2O$  are allowed to pass through c. The stirrer motion and wash water flow are then stopped, the pressure is released, and the apparatus is disassembled. The water layer above the catalyst is decanted and the solid transferred into a 250-ml. centrifuge tube by flushing with 95% ethanol. The material is washed three times by stirring (not shaking) with 95% ethanol (150 ml. each time) and the same number of times with absolute ethanol. If centrifuged after each washing at 1500-2000 r.p.m., one to two minutes are usually sufficient to settle out the product. The product catalyst is stored under absolute ethanol in a refrigerator. It cannot be stored indefinitely.



Fig. 337. Apparatus for continuous washing of Raney nickel in the absence of air. *a* water inlet; *b* water reservoir; *c* wash vessel; *d* discarded wash water; *e* hydrogen inlet; *f* nozzle for attaching a manometer; *g* waste water outlet.

All the above operations should be carried out as rapidly as possible. The time from the beginning of the run to the final placing of the catalyst into cold storage should not exceed three hours.

PROPERTIES:

Very dense grayish-black powder. Used as a hydrogenation catalyst [34]; it can also serve, just as Raney iron or cobalt, as the starting material for the production of the corresponding carbonyl compounds [19]. Also used as a catalyst carrier [9]; can be efficiently activated by treatment with metals of the platinum group [4].

The individual catalyst particles are very porous. The primary particle size ranges from 10 to 100 A [18, 42]. Raney Ni W-6 contains 12.7% Al [20]; its specific surface was determined as 87 m. $^{2}$ /g. [44].

GENERAL:

The intermediate stages of Raney's [33] general method for the preparation of catalytically active metal skeletons can be varied over a wide range, making possible products of widely varying activities. Let us discuss these individual stages.

The storting alloy is usually prepared by fusion of the components; this fusion should yield as homogeneous a structure as possible [39]. In another method, a fine powder of the pure catalytic metal is mixed with Al powder; the mixture is pressed into tablets and sintered for some time at moderate (of the order of 700°C) temperatures [46]. The Raney alloys can also be obtained by aluminothermic synthesis [11].

The optimum composition of the starting alloy, which determines the catalytic properties to some extent [36], is controlled by several considerations, namely: a) alloys with too high a content of catalytic metal yield products of low activity [6]; the upper limit of allowable active metal content varies from metal to metal. b) Catalysts obtained from alloys of differing active metal contents have somewhat different selectivities in the same reactions [6, 7]. c) Alloys with a definite composition such as NiAl may be so resistant to the leaching solvent used that no useful catalyst, results. The use of ordered solid solutions (e.g., Ni<sub>2</sub>Al<sub>3</sub>) does not offer any advantages, since the Ni atoms tend to regroup into an undesirable configuration after the Al is leached out [42].

Decomposition of the alloy should expose the skeletal metal structure. However, complete removal of the alloying metal (which accompanies the active one) requires drastic conditions and leads to products of poor activity [3]. For this reason the decomposition conditions are selected so as to leave some Al in R. WAGNER

the catalyst. The milder the conditions under which the decomposition takes place, the more active the catalyst and the larger the percentage of Al in the final product. In some case, it is sufficient to leach out the surface Al [40, 41; see also 34]. This also permits activation of the walls of the catalyst-containing reactor [38]. It is improbable that the Al in the catalyst is present as Al<sub>2</sub>O<sub>3</sub> [20, 45].

Skeletal metal	Second metal (weight %)	Leaching fluid	References
Fe	Al (20/80) Al	Aqueous NaOH Aqueous NaOH	[17, 30, 34] [15]
Co	Al	Aqueous NaOH	[5, 11, 12, 34]
Ni W-1 W-2 W-3, -4 W-5, -6, -7 W-8	A1 (50/50) A1 (50/50) A1 (50/50) A1 (50/50) A1 (50/50) A1 Mg (50/50)	Aqueous NaOH Aqueous NaOH Aqueous NaOH Aqueous NaOH Aqueous NaOH Aqueous NaOH Acetic acid	[8] [27] [32] [2] [25] [6, 29, 39] [28]
Cu	Zn Al, Zn (50/45/5) (Devarda's alloy)	Aqueous NaOH Aqueous NaOH	[24] [10, 26]
CoNi	Al (2/48/50) (5/45/50) Si (25/25/50)	Aqueous NaOH Aqueous NaOH	[35]

Literature	references	for	the	preparation	of	active	metals
by the Raney process.							

As far as the effect of the decomposition conditions is concerned, the following can be reported. The activity of the product is proportional to the rate of the decomposition and varies inversely with the decomposition temperature. The rate can be enhanced by starting with as fine metal powder as possible and adding the latter as rapidly as possible to the decomposing medium [6]. The primary particle size of the catalyst (as determined by x-ray analysis) always increases with the hydroxide concentration and the temperature [23]. However, the hydroxide concentration has little effect on the activity of the catalyst. Adkins [1] has presented a number of conclusions on the effect of these external conditions on the preparation of Raney Ni.

1628

#### MISCELLANEOUS CONSIDERATIONS

In addition to the Al, the ready Raney nickel catalyst also contains hydrogen, to which the pyrophoric nature of the product [22] is due. For the nature of bonding of the hydrogen, see [13, 14, 16, 37, 43]. The removal of hydrogen leads to loss of catalytic activity, which can not be restored by renewed treatment with hydrogen, even though the powder patterns of hydrogen-treated inactive preparations do not differ from those of active ones [22]. Since the catalyst must contain hydrogen, a special technique is required if it is to be used in deuteration reactions [25].

This general method for preparation of catalytically active structures is also applicable to metals other than Ni. It is also useful with alloys [31, 34]. In addition to Al, the alloy component which is leached out may be Si [12], Zn [24] and occasionally Mg [28] (dilute acetic acid is used as the leaching fluid). In determining the optimum composition of the Raney alloy for a specific purpose, one must also take into account the effect of the second metal.

Highly active skeletal Si was obtained by multistage removal of Ca from CaSi; the silicon metal was arranged in the form of a network consisting of six-membered rings [21].

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#### Nickel Formate-Paraffin Catalyst

## $Ni(HCOO)_2 \cdot 2 H_2O = Ni + 3 H_2O + CO_2 + CO_{184.8} 58.7$

Dry, precipitated NiCO3 is dissolved in a 20% stoichiometric excess of 50% formic acid (80°C). The salt that crystallizes on cooling is filtered and dried at 110°C. Then, 100 g. of this formate is placed in a 500-ml. round-bottom flask equipped with a 12-mm. I.D. condenser tube (method of Allison et al. [1]) and slowly heated in aspirator vacuum together with 100 g, of paraffin wax and 20 g. of paraffin oil. The evolving gases are washed three times with paraffin oil to trap entrained paraffin wax which might plug the tubing. The product is then held for one hour at 170-80°C to remove the water of crystallization, and then the temperature is raised to 245-255° to decompose the formate; the termination of the reaction after an additional four hours can be recognized by a decrease in pressure. The reaction product is poured onto a metal sheet while still hot. After cooling as much as possible, the top paraffin layer is scraped off and the remaining very black mass is broken up into coarse pieces. Immediately before use, these paraffin-coated pieces are treated on a large Buchner funnel with a large quantity of hot water to remove most of the paraffin. The residue is dehydrated with pure ethanol; it is then immersed several times in petroleum ether, removing the petroleum ether by suction.

#### PROPERTIES:

Loose, black, nonpyrophoric powder; relatively stable in air, provided the paraffin is completely removed and well wetted by water. Shows an activity level similar to that of Raney Ni in hydrogenation of aromatic nitro compounds in aqueous solutions.

#### GENERAL:

Thermal decomposition of some metal compounds whose anions are reducing agents gives the metal, which may have a very high catalytic activity. The active metal may then react with air or with gaseous reaction products. If air is not allowed to penetrate (the reaction is conducted in an inert liquid), and the nascent gaseous products are quickly removed (use of high vacuum or a stream of inert gas—see use of  $H_2$ , p. 1616), it is sometimes possible to obtain the metal in its active form.

The following starting metal compounds can be used in this procedure.

1632

Hydrides, for example, CuH [16],  $ZnH_2$  [15],  $CeH_3$  [5],  $UH_3$  [8] and so forth; these are decomposed in high vacuum at relatively low temperatures. Active uranium prepared from  $UH_3$  has the remarkable ability to absorb large quantities of  $H_2$ ,  $O_2$ ,  $N_2$ , CO,  $CO_2$  and other "base" gases; it can thus be used for the purification of "inert" gases, especially in closed systems [4].

Formates, for example, those of Co [17], Ni [2, 9, 17] and Cu [17]. Decomposition of these compounds yields a very porous active metal structure of crystallites; the same is true of Ni and Co oxalates [3, 13]. The decomposition of these salts is, to a large extent, a topochemical reaction, in which the nascent freemetal atoms regroup themselves within a very small region. To obtain these metals in the form of carrier-supported catalysts, one can start with a mixture of salts [3].

A number of other organometallic compounds among them several acetylides [7] and nitrides, give the metal in a more or less pure form on thermal decomposition.

Finally, active, sometimes even pyrophoric metals can be obtained by thermal decompositions of **amalgams**. Thus, fine powders of Be [14], Cr [6] and Ni [9, 10] are obtained from their amalgams upon removal of Hg by distillation. Since electrochemically obtained Fe and Co amalgams [10, 11, 12] decompose spontaneously, the active metal can be separated by simple mechanical means. This type of cobalt is an extremely active hydrogenation catalyst, while Ni produced from an amalgam is totally inactive [10].

Highly active noble metals may also be generated from other compounds. Thus, Pd, Ir and Pt "sponges" are obtained upon calcination of their ammoniumhexachloro complex salts (see p. 1562).

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## Active Copper

A fresh solution (800 ml.) of anhydrous Zn-free  $CrCl_2$  (80 g.) is prepared in water strongly acidified with HCl (see p. 1367). Carbon dioxide is bubbled through and the solution is cooled to 0°C. Then, an ice-cold solution of 60 g. of  $CuSO_4 \cdot 5 H_2O$  in 350 ml. of water is added with vigorous stirring. The reduction starts immediately and ends in a short while. The precipitated copper powder is washed several times by decanting with water. The water is then displaced with ethanol. The alcohol is, in turn, displaced with ether or benzene. The product may be stored under either of these liquids, unless it must be inactivated before use (see p. 1614).

**PROPERTIES**:

Extremely fine red powder with no metallic luster. Very useful as a catalyst in organic chemistry [16, 17].

GENERAL:

The usefulness of this homogeneous phase reduction is restricted to the more noble metals. The range of reducing agents which can be used here is quite wide: cations of low valence ( $Cr^{2+}$ ,  $Fe^{2+}$ ,  $Ti^{3+}$  and others), reducing anions ( $S_2O_4^{2-}$ ,  $PH_2O_2^{-}$ ,  $HCOO^{-}$ ), as well as  $H_2O_2$ ,  $N_2H_4$ ,  $NH_2OH$ ,  $H_2CO$  and so forth may all be employed.

These reductions require, as a rule, quite specific reaction conditions, especially as far as the hydrogen ion concentration and temperature are concerned. Both of the latter factors also affect the particle size of the product: the temperature does so in the way one would expect, but the pH in a less predictable fashion. Thus, for instance, gold precipitates from alkaline solutions in smaller primary crystallites than those obtained from acidic solution [8]. Certain reactions require above-atmospheric pressures [6, 10].

Homogeneous phase reductions have also been carried out in liquid NH<sub>3</sub> [1-5, 22-30]. The use of this solvent extends the general method to less noble metals, whose halides may thus be reduced with alkali metals. The intermediate compounds used are metal ammines; these are carefully decomposed frequently to give the active metal. The activity of the product increases with the atomic weight of the reducing alkali metal [26]; it is frequently maximum when Ca is the reducing agent [4, 5]. Many metals obtained via this procedure have remarkable catalytic activities. The use of the method is restricted by the tendency of many of the metals to combine with the reducing agent under the reaction conditions. Also, some of these metals undergo irreversible reactions with the solvent [2]. Solid solutions may be precipitated from solutions which contain two easily reduced cations (see below). Since the free energy of the less noble metals decreases as a result of formation of mixed crystals with more noble metals, the aqueous-solution procedure is not restricted to the alloys of noble metals. If the more readily reduced metal is also quite insoluble under the reaction conditions, it yields the nuclei upon which the remainder of the precipitate crystallizes. Such precipitates are, in general, finer and have a much narrower particle size distribution than chemically uniform materials [14. 24].

Metal	Reducing agent	Reaction medium	References
Cu	$Cr^{2+}$ S <sub>2</sub> O <sub>4</sub> <sup>2-</sup>	acid ammoniacal	[7, 16, 17] [9]
Ni	PH <sub>2</sub> O <sub>2</sub>	neutral; ammoniacal	[6, 10]
Ag	N <sub>2</sub> H <sub>4</sub>	ammoniacal	[19]
Pt (Pt black)	HCOO-	neutral	see p. 1562
Au	Н <sub>2</sub> О <sub>2</sub> Н	alkaline	[8]
	NH <sub>2</sub> OH	-	[20]
Ag-Au	Fe <sup>2+</sup>	acid	[14]
	Н₂СО	alkaline	[12]
Ag-Hg	H <sub>2</sub> CO	alkaline	[11]

Precipitation from aqueous solutions:

Metal	Reducing agent	References	Metal	Reducing agent	References
Fe Co Ni	K K K Li, Na, K, Rb, Cs	[23] [25] [5, 22, 24] [26]	Ru, Rh, Pd Ag Ir Pt	K Na, K, Ca K K	[29] [1, 3, 4] [28] [27]

Precipitation from solutions in liquid ammonia:

Homogeneous reductions may also be carried out in the **gaseous phase.** In this case, the product metal must form a sufficiently volatile reducible compound (chlorides are frequently useful in this respect) and the reduction temperature must not be too high. Suitable reducing agents are the vapors of easily volatile base metals or hydrogen:

> $SiCl_4 + 2Zn = 2ZnCl_2 + Si$  [15]  $TiCl_4 + 4Na = 4NaCl + Ti$  [18]  $VCl_4 + 2H_2 = 4HCl + V$  [13]

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Carbonyl Iron

 $\begin{array}{rl} \mathrm{Fe}(\mathrm{CO})_5 \ = \ \mathrm{Fe} \ + \ 5 \ \mathrm{CO} \\ 195.9 & 55.9 & 140.1 \end{array}$ 

## I. "FIBROUS" IRON

In the method of Beischer [1], a nitrogen stream saturated with  $Fe(CO)_5$  at some temperature is combined in the bulb-shaped reactor of the apparatus shown in Fig. 338 with a stream of very hot nitrogen. The hot-gas quantity is always several times that of the cold one. Thus, for example, if the carbonyl-saturated  $N_2$  flows at a rate of 2 liters/hour, the flow rate of the hot  $N_2$  must be 40-100 liters/hour. If the reaction temperature is maintained between 200 and 700°C and the Fe concentration in the decomposition zone does not exceed 10 mg./liter, a uniform, fibrous product collects in the settling chamber.

PROPERTIES:

Fibrous carbonyl iron, prepared at 200°C and at a reactor concentration of 1 mg. of Fe/liter, quickly absorbs 10-15% of its weight in  $O_2$  upon exposure to air; it must therefore be handled



Fig. 338. Preparation of very fine iron powder with fibrous particles. a settling chamber; r reactor (30 mm. I.D.); t thermocouple; p nitrogen heating tube (20 mm. diameter), filled with small porcelain pieces, and heated either electrically or with a series burner; s flowmeter; c carbonyl storage.

and used in an  $N_2$  atmosphere. The individual fibers reach a length of  $10^8$  A at a rather uniform thickness of about 2000 A. They consist of primary particles 70-90 A long (as determined by x-ray analysis).

#### **II. IRON GLOBULES**

In the method of Beischer [1] iron globules are formed at the maximum possible  $Fe(CO)_5$  concentration in the decomposition zone. The apparatus of Fig. 339 is used. The air is flushed out with a moderately fast stream of N<sub>2</sub> introduced via the inlet tube to *a*. Then the liquid carbonyl compound is vaporized at a rate of 30 ml./hr. and the vapor fed into the decomposition chamber, which is heated to 200-600°C (depending on the reaction conditions). At this point the N<sub>2</sub> flow is either reduced or shut off completely. The tubing from the distillation flask to the decomposition tube (which is surrounded by a vertical heater) must be well insulated or maintained at about 110°C by means of a small electric coil or tape in order to avoid decomposition of the iron carbonyl. The first crop of product does not have the desired properties. A uniform powder consisting of microscopic globules is obtained only after a certain induction period.

#### PROPERTIES:

The individual globules have a diameter of  $10^{4}$ - $10^{5}$  A and a peculiar structure similar to onion skin [5]. They contain (probably due to catalytic decomposition of CO on their surface) about

1% C, and they pick up 1-2% O<sub>2</sub> on exposure to air. The particles grow rapidly when heated to above 350 °C [4]. Because of its particle size carbonyl iron is useful in solid-solid reactions (e.g. in metallurgical sintering processes).



Fig. 339. Preparation of very fine iron powder with globular particles. *a* iron carbonyl distillation flask; *b* oil bath; *c* decomposition reactor and furnace; *d* thermocouple; *f* filter. GENERAL:

The process may also be used with other metals that form volatile, readily decomposed carbonyls. The optimum reaction conditions vary from case to case. In some instances, suppression of carbide formation is the major problem [2].

A material with a developed inner structure (i.e., porous) is developed if the decomposition is carried out completely in the reactor chamber (whereby the heat is supplied by radiation or hot gases). If the reacting gases are also made to follow a vortex path, then uniform, small particles are obtained [3]. However, any undecomposed carbonyl which reaches the wall of the vessel decomposes on it, precipitating the metal in the form of a tenaciously adhering mirror or in layers of leaflets (see p. Incomplete removal of air 1644). or deliberate addition of oxygen to the reactor may produce fine metal oxide aerosols (see p. 1669).

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## **Explosive Antimony**

A 25% solution of SbCl<sub>3</sub> in 10% hydrochloric acid is electrolyzed at 20°C in the apparatus shown in Fig. 340 [6]. The solution is

obtained by adding 300 ml. of conc. HCl to 500 ml. of water, dissolving 250 g. of  $SbCl_3$  in this mixture, and adding water to make up one liter. The electrolysis vessel, placed in a constant-temperature bath, is either a 650-ml. three-neck Woulfe flask or a one-

liter filter jar [7]. The anode a consists of very high-purity antimony; if it is short and does not protrude from the electrolysis jar, it is extended with a platinum wire, the remaining length being supplied with a copper wire. The cathode c is best prepared from a 10cm.-long piece of 1-mm.-diameter Pt wire, but copper [7] or manganin [18] wires of the same diameter are also suitable: the cathode c is bent to a U shape and rigidly attached to the stirrer (with the two free ends directed downward). The electrical connection is made via a drop of mercury placed inside the hollow stirrer shaft; this pool contacts a sealed-in platinum wire, which in turn makes contact with the cathode wire.

The electrolysis is started at a very low current; after five minutes the current is increased to give a cathode density of approximately 0.3 amp./in<sup>2</sup>, while stirring at a rate of 1000 r.p.m. Since the cathode surface gradually increases during the run due to deposition Fig. 340. Preparation of explosive antimony. *a* anode of pure antimony; *c* cathode of platinum wire.

of the smooth, shiny metal, the current must be gradually increased (25% in two hours). A two-hour run yields about 400 mg. of metal deposit, sufficient for demonstration purposes. After completion of the electrolysis the cathode is carefully removed (avoid bumping against the vessel wall), washed with conc. HCl, then with water, and rinsed with alcohol and ether.

PROPERTIES:

X-ray analysis [8, 10, 11] indicates that explosive antimony is amorphous, with a crystal structure almost like that of a liquid except for some short-range ordering of the atoms. This state is stabilized by small amounts of  $SbCl_3$  or SbOCl, which contaminate the precipitate. The electrical conductivity is 4-5 orders of magnitude smaller than that of the pure element [4, 18]. Crystallization, which is a first-order reaction [4], may be initiated by scratching the walls, by slight heating, or by an electric discharge. Due to a "heat of crystallization" of 2.5 kcal./g.-atom [5, 7], the trichloride is vaporized and becomes visible as a fog.

GENERAL:

Good cathodic deposition of metal powder or sponge is controlled by a number of factors which depend on the material itself and on the experimental conditions employed [14, 20, 21].

a) Cathode surface conditions. A low concentration of the deposited ion in the cathode surface film tends to prevent the formation of a solid deposit layer, as the crystals then tend to grow away from the surface. Low surface concentrations are enhanced by the use of a dilute electrolyte, complexes in which the metal is firmly bound, the presence of high concentrations of neutral salts, low solution temperatures, and high current density in an unagitated electrolyte.

b) Low overvoltage of the metal to be deposited; this causes needlelike and dendritic deposits which are easily crushed to a crystalline powder, e.g., Cd [19]. The overvoltage of the metals generally increases with complex formation and decreases with increasing temperature. These factors therefore work in a direction exactly opposite to that cited under (a).

c) Coprecipitation of basic salts from the cathode film also gives a porous precipitate, a condition favored by the use of neutral or weakly acid solutions (depending on the tendency of the metal ion to hydrolyze). It is, however, also possible to remove H<sup>+</sup> from the cathode film; this can be done by electrolytic deposition of electron-bearing ions (low cathode current yield)—a factor which may be enhanced by metallic impurities or low hydrogen overvoltage—or by oxidizing agents which use up H<sup>+</sup> while accepting electrons. The effect of temperature on this coprecipitation varies: higher temperatures favor hydrolysis, but also favor the increased supply of hydrogen ions from the solution by increasing the diffusion rate.

An appropriate selection and balance of these factors should permit electrolytic production of powders of every metal that can be deposited from aqueous solution. The particle size of the cathode deposit can be reduced by the use of ultrasound [2, 3].

The simultaneous evolution of  $H_2$ , which is not absolutely essential for the preparation of metal powder, nevertheless causes a certain fluffing of the metal precipitates, allowing them to occlude considerable amounts of nonmetallic impurities from the electrolyte. Thus Cu or Ag precipitates may, under proper electrolysis conditions, occlude several percent of citric or tartaric acids (or their salts), as well as asparagine, and so forth. In this state the powders have lower negative potentials than the pure metals [15], in the same way as explosive antimony [18].

Fused salts (fluorides, chlorides) easily yield metal powders [1] because they have virtually no overvoltage as long as the bath temperatures are kept low.

The following metal powders have been obtained by electrolysis of aqueous solutions: Fe [12, 20, 21], Ni [12, 20, 21], Cu [12, 16, 20, 21], Zn [12, 13, 20, 21], Ag [17], Cd [19, 20, 21], Sn, Pb [12], Ni-Pd alloy [9].

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## Silver

## (Active Agent for Reductors)

 $2 \text{ AgNO}_3 + \text{Cu} [\text{Zn}] = 2 \text{ Ag} + \text{Cu}(\text{NO}_3)_2 [\text{Zn}(\text{NO}_3)_2]$ 339.8 63.5 (65,4) 215.8

A sheet of electrolytic copper [7] or several zinc rods [2] are suspended in a well-stirred solution of 29 g. of  $AgNO_3$  in 400 ml.

of water acidified with a few drops of  $HNO_3$ . The reaction starts immediately. When the solution is free from Ag<sup>+</sup>, the copper sheet and the stirrer are removed and the silver sludge is washed several times with dilute sulfuric acid (decantation). This removes most of the copper (zinc); the sludge is then transferred to the reductor tube and further washed with dilute sulfuric acid until free from Cu (Zn). The sulfuric acid is then displaced with 1N HCl; the acid must always cover the silver in the reductor, whether it is being used or just stored. Any air bubbles present are removed by shaking.

The regeneration of the silver in the reductor proceeds via method of Wislicenus [9]. A small piece of zinc is placed on top of the silver filling of the column (this column packing is blackened for about 3/4 of its length by superficial chloride formation). The reductor should be filled with dilute sulfuric acid. The reduction of the AgCl proceeds rapidly if the Zn makes good contact with the reductor material.

PROPERTIES:

Fine silver-gray powder; used in analytical chemistry as a packing for Jones reductors (reductions in hydrochloric acid solutions).

GENERAL:

This process does not usually yield very pure products [4, 6]. The less noble metals (e.g., Cu [4]) become pyrophoric at low temperatures. The somewhat higher energy level of these preparations is due to the large surface area and to lattice defects. Some metals tend to form fibrous structures [8] with appropriate reducing metals and solvents.

Zinc can also be used as a reducing agent for the preparation of Cu [4, 5], Ni [3] and Sn [1].

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## Deposition of Metals from the Vapor Phase

Deposits of metals from the vapor phase [2, 5, 6, 10] are especially useful for studies of very pure materials, where it is desired to correlate the structure and the electronic state of a solid with its catalytic activity. Metal deposits can be investigated by electron diffraction, conductivity measurements, and optical and magnetic techniques even while they are covered with a layer of adsorbed material or while actually participating in a catalytic process.

The experimental apparatus varies with the type of study. In general, the parts of a glass apparatus should be fused together so as to avoid greased glass joints. This means that not only the reaction vessels proper but all the auxiliary devices such as gas receivers, cold traps, manometers and so forth must be fused to the apparatus prior to the start of a run. To obtain reproducible results, the catalyst carrier is carefully purified prior to sealing the apparatus, heated slowly to 400-500°C, and baked at this temperature for several hours in a high vacuum. Only then is the carrier temperature adjusted to the level required for condensation and the deposition of the metal vapor started. All these operations must be carried out in the vacuum of a running pump. Fresh metal deposits adsorb gases extremely readily. For this reason all gases other than those actually needed in the process (especially those which may be catalyst poisons) should be removed from the apparatus prior to the start of deposition.

**Reaction vessels:** The inside wall is usually used as the support for the deposit. The vessels, which are made of quartz or glass, are of two common types: spherical flasks for adsorption measurements [1, 29] and cylinders, which are especially suited for experiments in catalysis [5, 22]. Since the studies are usually conducted at constant temperature, the wall must have good thermal conductivity. The vessel is either immersed in a bath or surrounded by a jacket filled with a heat transfer medium.

Vaporization of the metal: The metal must be melted in vacuum. Aside from this requirement, which applies in all cases, the experimenter can choose from a variety of options. The usual procedure involves resistance heating of a suitably shaped wire coil. This method can be used where the metal has a sufficiently high volatility below the melting point to produce the vapor at an appreciable rate. In other cases it may be necessary to vaporize the metal at or above its m.p. In these cases, it is vaporized from the surface of a resistance-heated spiral or boat made of a high-melting metal (W, Mo or Ta). The current leads are sealed to the glass vessel; if necessary, they may be introduced in water-cooled ground joints. Suitable sheet-metal screens prevent condensation of metal in unwanted places.

Additional possible heating methods include high-frequency induction heating and cathode sputtering. However, one should remember that deposits obtained from a vapor and from a sputtered cathode differ somewhat in structure [14].

**Condensation**. In general, the condensation conditions greatly affect the secondary structure and the catalytic activity of the metallic deposit [7]. Depending on condensation conditions, the metal layers deposited on amorphous supports (glasses) at low temperatures may be either crystallographically disordered (random) or oriented [5, 25]. In an oriented layer the crystallites adhere to the support in a uniform fashion, the boundary with the support being the simply indexed crystal lattice plane that has the lowest atomic (or molecular) density. The crystallite arrangement in all other directions is totally random. Epitaxial growth may occur on crystalline material, with deviations of up to 15% in the lattice dimensions [8, 24, 34]. Metastable crystal modifications have also been observed in vapor deposits [11].

Condensation at low temperatures favors the formation of homogeneous mirrors, which remain stable at room temperature. Under these conditions mirrors are formed even by metals that otherwise would appear dull [28]. The specific surface area, which is controlled primarily by the melting point of the metal, is slightly higher in deposits obtained at low temperatures than in those produced at 0°C [32]. Higher condensation temperatures or the presence of inert gases (noble gases can be used in all cases, nitrogen sometimes, while hydrogen is completely unsuitable; see [12]) results in a decrease of the orientation of the crystallites (the latter cease to be oriented at sufficiently high condensation temperatures). An additional phenomenon appearing at high condensation temperatures or in the presence of gases is that the individual crystallites become smaller while retaining their normal lattice constants [20] and the deposits become dull black. In contrast to mirrors, they show a strong small-angle x-ray scattering [9].

Simultaneous deposition of lattice-distorting substances (NaCl,  $H_2O$ ) has been recommended for obtaining defect structures [12].

Metal deposits which are useful for preparative purposes may be obtained by thermal decomposition of suitable volatile metal compounds (hydrides, carbonyls) on hot surfaces. For instance, decomposition of Ni(CO)<sub>4</sub> on Pyrex glass wool at 150 °C produces a deposit of very finely divided nickel, which is an excellent catalyst for gas-phase hydrogenation of olefinic double bonds [4].

#### 1. ADSORBENTS AND CATALYSTS

Metal	Studies on the properties of the deposit	Adsorption measurements	Catalytic studies
Fe	[13, 14, 32]	[3, 21, 33]	[18, 27]
Co	[13]	[21]	
Ni	[5, 7, 11, 13, 20, 25	[5, 7, 26, 29, 33]	[5, 15, 16, 17, 18,
	32]		22, 23, 27, 30]
Cu	[9, 13, 19, 28]	[1]	
$\mathbf{R}\mathbf{h}$	[32]		[16, 17, 18]
Pd			[16, 17]
W	[32]	[31, 33]	[17, 18, 27]

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### HYDRATED OXIDE GELS

The hydrated oxides used as adsorbents are called, more precisely, xerogels\* or aerogels. They are prepared by drying the corresponding hydrogels. The adsorptive capacity of the xerogels depends on the drying process used. The starting hydrogels are prepared by the following methods:

### PRECIPITATION REACTIONS

The molecules of precipitated hydrogels are usually arranged in a random order. However, ordered structures may form if the conditions are such that basic salts or aggregated anions can be produced during the precipitation or when such starting materials are used. The surface properties and pore structure of these gels, which show aggregates of submicroscopic globular particles under the electron microscope, depend on the precipitation reaction used. Among these reactions are:

1) Precipitation with an acid or a base (see hydrated chromium oxide gel, p. 1648; silica gel, p. 1648).

<sup>\*</sup>Translation Editor's Note: In American practice the term "xerogel" is little used, although more precise. We prefer "resin" for organic materials and "dry gel" for inorganic ones.

2) Hydrolyses of alkoxides that give preparations free of electrolytes (see aluminum hydroxide gel p. 1652).

### REACTIONS OF SOLIDS

These reactions should be topochemical and may use preformed starting materials. Among the preparative methods of this class are:

1) Topochemical hydroxide formation [see "glimmering" iron (III) hydrated oxide, p. 1654].

2) Leaching out one component of a solid mixture (see p. 1656). The preparation of gels by controlled coagulation of sols is

used more rarely. However, it may still be of interest in special cases. For instance, globular gel particles, which are useful as fluidized bed catalysts, are obtained by allowing the sol to drop into a medium which causes its spontaneous coagulation.

On simple drying in air or heating (if necessary), gels obtained by precipitation reactions show a much higher shrinkage than those obtained by reactions of solids. The shrinkage may, however, be avoided to a large extent if instead of drying the material one displaces the water with organic solvents of low surface tension. These are then removed by evaporation. In general, the pore volume increases as the surface tension of the liquid used decreases [7].

Aerogels with an especially high surface activity and specific surface (up to 800 m<sup>2</sup>/g. [4]) are obtained via the procedure of Kistler [5, 6]. In this case, the water in the hydrogel is displaced with an organic liquid which is miscible with water. The preparation is then heated in an autoclave to a point above the critical temperature of the solvent where the supercritical vapor is released. Gels prepared by this method retain the volume and the structure that existed prior to drying. In the preparation of hydrated oxide gels, one must remember that the reaction products age rapidly on contact with the aqueous mother liquor; in addition the aging process is markedly affected by the electrolyte content of the solution. Only a few precipitates remain amorphous for any length of time; most convert fairly rapidly to crystalline hydroxides or hydrated oxides [2, 3]. However, their gel nature initially remains almost the same. Some hydrated oxide gels must not be heated above a certain temperature during drying to avoid spontaneous crystallization with glimmering deflagration (compare also metamictic minerals [1, 8, 9]).

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# Hydrated Chromium Oxide Gel

 $2 \operatorname{Cr}(\mathrm{NO}_3)_3 + 6 \operatorname{NH}_3 = \operatorname{Cr}_2 \operatorname{O}_3 \cdot (\operatorname{aq}) + 6 \operatorname{NH}_4 \operatorname{NO}_3$  $(9 \operatorname{H}_2 \operatorname{O})$ 800.4 152.0

Slightly more than one liter of  $0.12M \text{ NH}_3$  solution is added with efficient stirring to one liter of  $0.04M \operatorname{Cr}(\operatorname{NO}_3)_3$  (the excess of NH<sub>3</sub> favors the formation of a floculent precipitate). The precipitate is washed 10 times by decantation with water, filtered and dried at 150 °C [5].

PROPERTIES:

Dark-green glasslike granules. The N<sub>2</sub> absorption isotherm gives a specific surface of  $310 \text{ m}^2/\text{g}$ . (BET method) [5]. Used as a hydrogenation and dehydrogenation catalyst [18].

# Silica Gel

To start with, 3.4 liters of sodium silicate solution (sodium waterglass,  $d^{20}$  1.37) is diluted with one liter of water (mechanical stirring). Then, 10N HCl is added at a rate of 10 ml./min. until thymol blue shows an acid reaction (pH 2-2.8). (After addition of 400 ml. of the acid the mixture becomes viscous and rubberlike. The acid addition is interrupted and the mass is broken up. It is then manually stirred while acid is added in drops. The mixing is continued until a thin suspension is obtained. The remainder of the acid is then added at the original rate until the desired pH is reached.) The mixture is then stirred for two additional hours at room temperature, suction-filtered and washed until the wash liquid is no longer acid. The gel is dried at 200°C for 12 hours, ground to the desired particle size, and finally washed free of Cl<sup>-</sup>.

#### PROPERTIES:

Dull-white gel granules; hardness approximately that of glass; high specific surface (500 m<sup>2</sup>/g.). Gels prepared according to the above directions are especially useful for chromatographic purposes [11].

#### GENERAL:

Many other hydroxide precipitates may be obtained in a way similar to that used for the hydrated chromium oxide gel. However, when selecting the starting metal salts one must remember that anions of multivalent acids, especially  $SO_4^{2-}$ , are frequently difficult to remove from the product by washing. For this reason nitrates, chlorides and perchlorates are preferred. In addition, hydrated oxides obtained from sulfates frequently have a very high tendency to spontaneous deflagration. Isomeric hydrated salts, such as chromium (III) chloride hydrates, may give a variety of hydroxide precipitates ([10]; see p. 1345). The base is usually a freshly prepared, carbonate-free NH<sub>3</sub> solution. In cases when the metal ion forms stable ammine complexes which make a quantitative precipitation difficult, the precipitation may be carried out with ammonium acetate or tetramethylammonium hydroxide solutions [22]. In addition to these agents and the various alkali metal hydroxides, active MgO and alkaline earth hydroxides may also be used as precipitants. The impurity cations that are adsorbed on the precipitate may, under certain circumstances, act as catalyst or adsorption promoters. Hydrated oxide gels of amphoteric metals may also be obtained by careful neutralization of the corresponding alkali hydroxometallate solutions. As a rule, such precipitates are microcrystalline and contaminated with large amounts of alkali ions. Hydrated oxides of cations with a high ionic potential, such as Ti<sup>4+</sup>, may also be obtained by hydrolysis of their salts, whereby a dialyzer may also be used [28, 33, 34].

The quantity of the precipitating agent has a large effect on the properties of the product gel. Incomplete precipitations (final acidic solution in the case of hydrated metal oxide gels, final alkaline solution in the case of silica gels) produce soft, friable, strongly opalescent gels with a wide range of pore sizes and low specific surfaces. An excess of the precipitating agent, on the other hand, usually causes the condensation reaction to go to completion. As a result, the gel products are thoroughly crosslinked and the presence of the three-dimensional network makes them hard, elastic and translucent, with fairly uniform pore size. The properties of the product gels depend also on the reaction temperature, as well as the order and rate of addition of the reagent solutions. If the precipitating agent is added slowly to the salt solution, there is a possibility of forming an intermediate basic salt [12, 24]; these may then precipitate as well defined compounds [6]. On the other hand, slow addition of the precipitant may yield isopolyanions (polysilicate ions and so forth); these last may also be added beforehand to obtain special effects. Such precipitation products are aged in a specific way compared with gels obtained by fast addition of the salt solution to the precipitant [17, 20].

The adsorptive selectivity of the gels may be influenced to a certain extent by the preparative conditions. One may, for instance, produce silica gels which adsorb a specific dye of characteristic molecular shape and charge distribution. This is done by dispersing this dye in the silicate solution and the precipitating the gel in the presence of the dye [2, 4, 9]. Similar experiments have been carried out with optically active compounds ([3], see also [1]). The silica gel surface may also be modified in a specific way by adsorbing on it appropriate substances; this yields preparations with completely new adsorptive properties [14, 19].

The adsorptive activity of gels is reduced not only by heat but also by grinding. Grinding produces a slight reduction in the specific surface. This is occasionally accompanied by a reduction of the average pore radius [16].

The above general remarks apply also to mixed precipitates of metal hydroxides as well as to silica-hydrated metal oxide mixed gels. As would be expected, products obtained by simultaneous precipitation of two (or more) compounds differ from those obtained by mechanical mixing of finished gels; both of these types of gels are in turn different from mixed gels produced by sequential precipitation in the same solution. Finally, we should mention the so-called chalky silica gels (see p. 1656).

To summarize, the quality of the final product gels depends on the history of the preparation. In view of the many possible slight variations in the procedure which affect the reactions involved, it is not surprising that the products vary in quality.

The product gels may be freed from impurity ions (usually present in large amounts) by dialysis or electrodialysis. However, even this procedure does not yield gels completely free from electrolytes. If completely pure products are desired, it is best to use the hydrolysis of alkoxides presented on p. 1652.

Finally, let us cite a number of new publications dealing with the preparation and testing of hydrated oxide gels: hydrated aluminum oxides [30, 32]; silica gels, unmodified [11, 13, 16, 21, 27]; silica gels, modified [1-4, 9, 14, 19]; hydrated titanium oxides [8, 15, 28, 33, 34]; hydrated chromium oxides [5, 10, 18, 23-25]; hydrated iron oxides [7, 17, 20, 23, 29]; hydrated zirconium oxide [26]; hydrated tin oxide [31]; hydrated thorium oxides [8].

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### Aluminum Hydroxide Gel

# $\alpha$ - GEL BY THE METHOD OF WILLSTÄTTER AND KRAUT

### $Al(OC_2H_5)_3 + 3H_2O = Al(OH)_3 + 3C_2H_5OH$

In the method of Schmäh [25], 400 ml of  $CO_2$ -free doubledistilled water is placed in a one-liter, three-neck flask, whose center neck carries a mercury-seal stirrer. One side neck is closed off with a soda lime tube. The other is fitted with a dropping funnel with an ungreased stopcock from which a solution of 3 g. of aluminum ethoxide in 200 ml. of absolute ethanol is allowed to run in a thin stream into the vigorously stirred water (the aluminum ethoxide is prepared from Si- and Fe-free 99.99% Al by one of the methods given on page 840 and is then dissolved by refluxing with the required amount of ethanol). During the hydrolysis of the ethoxide, the temperature rises by 6-8°. The precipitate is washed by decantation with double-distilled water (the settling may be speeded up by centrifugation).

PROPERTIES:

The  $\alpha$ -gel is completely free of electrolytes; its surface is quite alkaline. The fresh gel is completely amorphous [19]; it ages rapidly to bayerite via the intermediate stage of böhmite, the alkalinity of the surface decreasing considerably in the process [16, 17]. Aluminum oxides prepared from this gel are more active than the usual aluminum oxide catalysts [1].

#### GENERAL:

This method of preparation allows some latitude in the hydrolysis conditions as well as in the alcohol moiety of the alkoxide. The hydrolysis may be carried out in absolute ethanol solution, the hydrolysis agent being atmospheric moisture (stirring in air). Alternatively, aqueous alcohol may be added to the ethanol solution, or the hydrolysis may be carried out in the reverse fashion, by addition of the alcoholic solution to water (the water may be hot, if required). Ammonia solution may be used instead of water. Solid alkoxides (such as some methoxides) may be hydrolyzed in a stream of moist air with heating if required [27]. Since the hydrolysis of alkoxides of metals of variable valence may proceed in steps [2, 5, 11], the most active products are obtained on fast precipitation.

The ability of the alkoxide to undergo hydrolysis depends on the nature of the alcohol moiety. This ability decreases with increase in the molecular weight of the organic part and increases in the order of primary to tertiary alcohol (as shown by zirconium and titanium alkoxides [6, 8, 26]). The volatility of the alkoxides increases in the same order. This fact is of some importance, because with elements having a high atomic number, it is often only the tert-alkoxides that can be distilled (and thus purified), even in high vacuum (for example, Th [10]).

A high hydrolysis temperature accelerates the aging process, so much so that in some cases the amorphous hydrated oxide cannot be isolated.

Metal	Method of preparation of the alkoxide	Hydrolysis and products of hydrolysis
Al Si Cr Fe Zr Nb Sn Ce Hf	<pre>p. 840 this handbook [7], see p. 702 [4, 7, 11, 24, 26, 27] [30] [24, 28, 29] [6-8, 24] [14] [24, 31] [13] [9]</pre>	[1, 3, 16, 17, 19, 21, 25] [2, 18, 22, 32], see p. 698 [5, 11, 15, 18, 20, 23, 26, 27] [30] [29] [6] [31]
Ta Th	[12] [10, 13]	

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# "Glimmering" Hydrated Iron (III) Oxide

In the method of Kohlschütter et al. [3], 20 g. of  $FeSO_4 \cdot 7 H_2O$ (analytical grade) is boiled for 40-60 minutes in a Kjeldahl flask with 200 ml. of pure 70%  $H_2SO_4$ . The water vapor is allowed to escape in order to concentrate the acid. Following the reaction the tablet-shaped crystals (which range in size up to 2 mm.) are collected on a fritted-glass filter. They are washed with some water, then several times with acetone, and are then dried in vacuum. The dry crystals are reacted (vigorous stirring) with 200 ml. of 2N aqueous ammonia (or correspondingly less of a more concentrated solution). The reaction is over in 10-15 minutes. The product is allowed to settle, the mother liquor is decanted, and the residue washed 4-5 times by decantation with water. The hydrated oxide is collected on a filter, rinsed several times with water, and dried with acetone and ether.

To obtain a preparation with a particularly impressive glimmer, the substance is predried with ether (as above) and then carefully heated at  $300^{\circ}$ C for 30 minutes.

#### PROPERTIES:

Contains, even after baking at high temperature, a considerable amount of water. Single, freely flowing particles retaining the external shape of sulfate crystals. Amorphous on x-ray analysis; the hydroxide framework is permeated by numerous pores. Crystallizes spontaneously and with glowing to  $\alpha$ -Fe<sub>2</sub>O<sub>3</sub> on heating to 350 °C.

#### GENERAL:

The ability of the system to undergo a topochemical reaction depends on several prerequisites: the solid should not dissolve too rapidly (otherwise the reaction does not occur at the interface but in solution). This prerequisite is often fulfilled by sulfates of trivalent metals. If necessary, the rate of solution may be reduced by adding a sufficient quantity of an organic substance to the aqueous solution. The organic substance may be acetone, glycol, glycerol, dioxane and so forth, or in the case of hydroxide "precipitations," it may be pyridine, mono-, di- or triethanolamine, morpholine and so forth (see [6]).

The reaction product, which is formed on the surface of the reacting crystals, should not form a solid film: it must be permeable to all reagents present in solution so as to allow the completion of the reaction throughout the individual crystallites.

Instead of simple salts, one can use double salts with readily soluble components. Thus,  $KAl(SO_4)_2 \cdot 12 H_2O$  yields a granular hydrated aluminum oxide with a very porous structure [6]. Basic salts may also be reacted topochemically to yield hydrated oxides.

The 'glimmering' was also observed in hydrated oxides of Cr, Ti, Zr [1, 10], Sc, Nb, Ta [10] and others. Another topochemical method for preparation of hydrated oxides consists of an extension of the procedure proposed by Raney in the case of metals. The required starting mixtures may be obtained by fusion of the components or by coprecipitation. A commercial fusion procedure yields Vycor glasses with interesting adsorptive and catalytic properties [4, 5, 9]. The fusion procedure is also used to obtain sodium ferrite-aluminate mixed crystals; these then yield hydrated iron (III) oxide skeletal structures showing considerable chemical activity (reacting with hot aqueous NaOH) provided the starting mixture contains an excess of aluminate [7]. The activity of such structures is due both to the small particle size of the product and to a strongly distorted crystal lattice due to "frozen" thermal vibrations [8].

Coprecipitation of silicic acid with hydrated oxides of metals such as Fe, Al, Cr, Ca, Cu, Ni and so forth yields silica gels that, after washing, drying and activation by leaching with hydrochloric acid, give chalky materials (provided the metal concentration in the starting mixtures is high [2]).

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# ACTIVE METAL OXIDES

The methods of preparing active metal oxides may be grouped according to similarities of procedure or of structure of the product.

### PREPARATION BY TOPOCHEMICAL REACTIONS

The topochemical reactions that give activated oxides are essentially thermal decompositions of suitable compounds:

1) Dehydration of hydroxides [see aluminum oxide, p. 1660;  $\alpha$ -iron (III) oxide, p. 1661].

2) Thermal decomposition of nitrates, carbonates, oxalates and so forth (see magnesium oxide, p. 1663; zinc oxide, p. 1664).

3) Dehydration of hydrated oxides in the mother liquor [see lead (IV) oxide, p. 1668].

The products prepared under mild conditions are usually (porous) pseudomorphs of the particles of the starting material (see [17]). Reproducible preparation of active oxides requires not only constant decomposition conditions, but also consistently uniform starting materials (constant conditions during precipitation and so forth). The effects of the chemical composition and the physical structure of the starting material on the properties of the final product decreases with increasing decomposition temperature. The extent of this decrease is proportional to the extent of ageing of the products during reaction conditions.

Kinetic measurements on thermal decomposition reactions yield several empirical equations. In general, these may be interpreted as follows:

1. If the order of the reaction is 1/3, the process is diffusion-controlled.

2. If the order of the reaction is 2/3, the process is a decomposition reaction which progresses from the outside to the center.

If the specific surfaces of the products are plotted against the decomposition temperature (at constant decomposition time), typical curves are obtained [3]. From these, once can not only read off the optimum conditions for creation of maximum surface but one can also obtain some indication of the mechanism of the decomposition reaction. Complete explanation of these phase relations (in some cases, an extremely complicated problem) requires the use of x-ray and IR spectroscopy and thermogravimetric methods.

In practice the decomposition temperature must be exactly maintained. Thus one cannot employ a shorter reaction time at higher temperature without incurring a loss of activity in the material. The effect of the atmosphere in which the reaction proceeds is especially remarkable. In some cases it was possible to reduce the activation energies for the decomposition by 10-15 kcal./mole (by comparison with those needed in air or vacuum) by proper choice of the gaseous atmosphere. In many cases, a properly chosen gaseous atmosphere allows the decomposition to proceed at unusually low temperatures; this in turn gives high active preparations [2].

PREPARATION BY CONDENSATION FROM A HOMOGENEOUS PHASE

The only oxides that can be condensed from the gas phase are those that are vaporized products of a chemical reaction occurring immediately prior to the condensation. Otherwise, vaporization of oxides is usually impossible due to the high boiling points involved.

Colloidal suspensions of oxides in air (smokes, see p. 1669) usually are very nonuniform because of the randomness of nuclei formation [1, 6]. The individual particles are usually globular. The growth of particles from the gas phase occurs in two stages:

Primary growth, which is exceedingly rapid and produces particles 500-1000 A in diameter. In this stage, the only material condensed is that which finds itself within the confines of a "sphere of influence" of a nucleus.

Secondary growth, which is much slower and depends mainly on the vapor pressure of the material under the temperature conditions prevailing during growth. This stage, corresponding to the aging of precipitates under mother liquors, can be suppressed by quenching of the material.

# PREPARATION BY INTERFACE REACTIONS

Under some conditions, the texture of oxide growth layers is determined by epitaxy with the support. Such layers are obtained by:

1) Surface oxidation of metals and alloys (e.g., bluing layers by heat treatment).

2) Oxidation of metal layers deposited electrolytically or from a vapor on carriers other than the metal itself. Occasionally, such surface layers show oxide modifications which are not known to occur in the pure material. In some metals the colored layers formed by strong oxidation are covered with needles or leaflets of the oxide [9]. The dimensions of these depend on the duration and the temperature of the oxidation. The rough surface thus produced may sometimes offer twice the normal specific surface [11].

Because these processes are related to catalysis, they have recently been studied by many investigators. This is especially true of the initiation reaction and the process kinetics [5, 7].

In view of the enormous complexity of the subject, the methods for the preparation of mixed and carrier-supported oxide catalysts can only be summarized here. They include:

1. Mechanical mixing (dry, in an atmosphere laden with moisture, under special gases, in suspension, and so forth [10]).

2. Coprecipitation, especially when solids of definite composition are desired; it is also applicable to the more rarely used solid solutions.

3. Precipitation of one component onto a carrier which is itself suspended in solution.

4. Adsorption of ions of one component from a solution and so forth [8].

The interaction of the components of a mixed catalyst (promotion of catalytic activity), which in general requires thermal activiation for development of the full effect, is a very complex process; occasionally, it yields very active and very specific catalysts.

The two special effects accompanying such an interaction are:

1. Alteration of the semiconductor properties of an oxide by inclusion of other oxides with different valences in the crystal lattice [5, 16]; this may also alter the activation energy of a catalytic process [14].

2. Adaptation of valence. The metal in the precipitated oxides of some transition metals adapts to the valence of the metal in the oxide carrier. Thus, transition metals preferably deposit on MgO in divalent forms while on  $\gamma$ -Al<sub>2</sub>O<sub>3</sub> they are trivalent and on TiO<sub>2</sub> (rutile) tetravalent [15]. This is because the oxide precipitate attempts to continue the crystal lattice of the carrier.

Various methods are available for shaping the oxide catalysts [4]. The following methods are used in laboratories:

The material, which may be moist if required, is pressed into a sheet. After drying, it is broken up and sieved. However, the high pressure applied during sheet forming produces a large decrease in the average pore size. This may lead to a considerable reduction of the catalytic activity of such preparation, especially at large reagent throughputs (the diffusion, which controls the overall process rate, becomes hindered by the small pore size [12]).

In the second method the dry powder is made into a paste with 80% ethanol. The paste is then rolled (without applying any high pressure) into a thin sheet, which is thenforced through a polished copper or nickel screen of suitable mesh size (1-2 mm.). After drying, the granulated material may be scraped off from the reverse side of the screen. If the dilute ethanol does not yield stable granules, the paste may be made with a saturated aqueous solution of the corresponding metal nitrate. In this case, the granules are shaped as above, dried and baked at 200-220°C until nitrous fumes cease to evolve [13].

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# Aluminum Oxide

Depending on the conditions, certain aluminum hydroxides yield active  $\gamma$ -oxides on thermal decomposition. These possess interesting adsorptive and catalytic properties.

I. ADSORBING AGENT

$$2 \operatorname{Al}(OH)_{3} = \operatorname{Al}_{2}O_{3} + 3 \operatorname{H}_{2}O_{156.0}$$
 101.9

An aluminum oxide which is especially suitable for use as an adsorbing agent is obtained by heating aluminum hydroxide gel (see p. 1652) or hydrargillite (see p. 820; for preparation of an almost completely alkali-free material, see [18]) for several hours at 250-300 °C. The heating proceeds in vacuum or in a stream of dry gas, and is continued until the concentration of the water has bound in the crystals decreases to 6-8 wt. %.

### PROPERTIES:

Fine, white powder which flows like sand; specific surface:  $250-300 \text{ m.}^2/\text{g.}$  [26]. The powder pattern shown böhmite lines, although the water content differs significantly from that of böhmite (15.02%). The thermodynamic potential, amounting to several kcal./mole, is due to the large surface and the defect lattice [27].

# II. CATALYSTS

Aluminum oxide catalysts are prepared by heating hydrargillite at 550-650 °C (other aluminum oxides give less active preparations); the content of water of crystallization is 1% or slightly less.

#### PROPERTIES:

Hygroscopic powder, very similar in appearance to the abovedescribed adsorption agent. The particles show a honeycomb-like secondary structure; the preparations show very uniform powder patterns (according to Glemser [17], they consist of  $\epsilon$ -phase). Any water adsorbed during use is incorporated in the form of OH groups and thus causes rehydration [18].

# a-Iron (III) Oxide

# For chromatographic adsorption

In the method of Glemser and Rieck [19] a solution of 1000 g. of  $Fe(NO_3)_3 \cdot 9 H_2O$  (A.R. or pure) in 2.4 liters of water is added with constant stirring to 2.4 liters of 6% aqueous ammonia. The precipitate is centrifuged off and washed with water until the wash liquor is free of nitrates. It is then dried at 50 °C. After two days at this temperature it is broken up and freed of dust on a U.S. 400 standard screen. Long heating (10-16 hours) of this crude product at high temperatures gives preparations with other activities. The maximum activity is usually reached by heating at 180-220 °C for 10 hours.

#### GENERAL:

The rate of dehydration of hydroxides and hydrated oxides may be varied within limits by choice of suitable experimental conditions. Aside from the obvious effect of temperature, the following dehydration procedures are open to the experimenter and give better products than does simple heating in open dishes or crucibles:

1. Removal of the water by means of a dry stream of air or other gas. The dehydrating action of the various gases is quite specific; hydrogen is an especially efficient dehydrating gas [25].

2. Reduction of the vapor pressure of water in the atmosphere surrounding the solid by operating under vacuum or using drying agents. If a drying agent is used, the material may be placed in a drying pistol instead of a desiccator. In this case, the sample may be heated to a high temperature while cooling the drying agent. This produces a large water vapor pressure gradient. It should be remembered that the pressure of water in equilibrium with active oxides is much lower than that encountered in equilibrium with inactive materials. In many cases it is impossible to obtain completely dry oxides without reducing the activity to a level considerably below the maximum. R. WAGNER

One remarkable phenomenon, which has been proved over and over again, is the fact that dehydrations of the above type do not alter the shape and size of secondary particles, while the size of the primary particles is subject to sharp variations [29].

Dehydration in hydroxides and hydrated oxides is often quite sensitive to impurities [1]. To obtain reproducible results, one should always use starting materials of the same purity.

Literature references for the preparation of active metal oxides by dehydration of hydroxides:

Oxide	Starting material	References
BeO	α-Be(OH) <sub>2</sub> (see p. 894)	[9]
MgO	Mg(OH) <sub>2</sub> (see p. 912)	[10, 23]
$\gamma - Al_2O_3$	Al(OH) <sub>3</sub> , amorphous (see p. 1652)	[24, 26]
	Al(OH) <sub>3</sub> , hydrargillite (see p. 820)	[2, 18, 20, 26, 27]
	Al(OH) <sub>3</sub> , bayerite (see p. 821)	[4, 18, 20, 27]
	AlOOH, böhmite (see p. 821)	[12, 13, 18, 26, 27, 29]
TiO2	Hydrated titanium (IV) oxide	[21]
Cr <sub>2</sub> O <sub>3</sub>	Hydrated chromium (III) oxide, amorphous (see p. 1648)	[3, 25]
α-Fe₂O₃	Hydrated iron (III) oxide, amorphous	[6, 8, 14, 19, 22]
γ-Fe₂O₃	$\alpha$ - FeOOH (see p. 1499) $\gamma$ - FeOOH (see p. 1500)	[6] [15, 16]
NiO	Ni(OH)2 (see p. 1549)	[28]
ZnO	$\epsilon$ -Zn(OH) <sub>2</sub> (see p. 1074) Other crystalline hydroxides	[5] [11]
CdO	Cd(OH)2 (see p. 1097)	[7]

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### Magnesium Oxide

Active MgO is prepared by calcination of the basic carbonate (for preparation, see p. 911). The following particle sizes have been observed, depending on the temperature and the duration of the calcination, [12]:

	Heating	Particle size in A					
Temp., °C	time, hr.	Determined by x-ray analysis	Observed by ultra- microscope				
370	60	30	100				
725	14	46	100				
820	14	50	100				
920	14	300	200 300				
980	14	300	300 500				
1200	4	300	10003000				

R. WAGNER

Since commercial basic magnesium carbonates are not welldefined compounds, the properties of the oxides vary to some extent. To obtain preparations with reproducible properties it is better start with well-defined compounds such as fine crystalline  $Mg(OH)_2$  (for preparation, see p. 912) or with  $MgCO_3 \cdot 3 H_2O$ [3, 22].

### Zinc Oxide

$$ZnC_{2}O_{4} \cdot 2 H_{2}O = ZnC_{2}O_{4} + 2 H_{2}O$$

$$189.4 153.4$$

$$ZnC_{2}O_{4} = ZnO + CO_{2} + CO$$

$$153.4 81.4$$

A solution of 27.3 g. of anhydrous  $ZnCl_2$  is in 200 ml. of water and 2.5 ml. of 2N HCl is prepared. Another solution, containing 31.3 g. of  $(NH_4)_2C_2O_4 \cdot H_2O$  in 2.0 ml. of water and 2.5 ml. of 2N aqueous NH<sub>3</sub> solution, is prepared separately. Both solutions are heated to 70°C, and the oxalate solution is then poured in a thin stream into the vigorously stirred zinc salt solution. The oxalate precipitate is washed by decantation with water until it is free of chlorides. It is then placed on a filter and dried by suction. The  $ZnC_2O_4 \cdot 2 H_2O$  is then transferred to a flat pan which is placed in a drying oven. The temperature is then raised to 240°C over a period of 6 hours and is then maintained at this level for an additional 12 hours. This treatment removes nearly all of the water of crystallization. The anhydrous oxalate is then converted to ZnO by heating at 400°C for 4 hours.

#### PROPERTIES:

Fine, white powder; untamped bulk density (pouring into a cylinder) 0.85 g./ml. The primary particles, whose lattice still contains defects, are larger than 500 A [18].

GENERAL:

Oxides of metals exhibiting low basicity may be obtained by thermal decomposition of their salts with volatile or readily decomposed acids. Such salts include nitrates, carbonates, formates and oxalates. In order to obtain active preparations, the decomposition conditions should be as mild as possible. The decomposition proceeds faster in vacuum than in air, so that the reaction temperature may be lower (for the same yield per unit time). An even stronger influence on the reaction rate is sometimes exerted by an appropriate gaseous atmosphere (see below).

#### NITRATES

Hydrates of most heavy metal nitrates have low melting points. Thus, on heating, they liquefy, decompose and leave sintered or foamed oxides of low surface area. The product may sometimes be improved by starting with lower hydrates or basic nitrates. Nitrate decomposition is important primarily in the production of carrier substances.

In compounds possessing several oxidation stages (e.g., elements of the manganese series), nitrate decomposition always yields the highest oxide possible at the given reaction temperature.

#### CARBONATES

Vacuum decomposition of carbonates is often used to obtain metal oxides where the metal is at the oxidation stage corresponding to that in the starting carbonate. However, with air present, at least partial oxidation is possible, sometimes even after the material has been cooled to room temperature. In addition to carbonates and their hydrates, as well as basic carbonates, this method is suitable for the decomposition of double carbonates where ammonium is one of the cations [for example, MgCO<sub>3</sub> · (NH<sub>4</sub>)<sub>2</sub>CO<sub>3</sub> · 4 H<sub>2</sub>O]. When the conditions are mild, very fine oxide powders are frequently obtained.

The decomposition of magnesite has been studied very thoroughly and proves to follow 2/3 order kinetics (see p. 1657 and [17, 21]). The effect of a gaseous atmosphere on the kinetics of this reaction consists mainly in the change in the activation energy for decomposition which it causes [8, 9].

Atmosphere	Dry air	Vacuum	Η₂	Moist air	НгО
Activation energy in kcal./mole	42.15	36.13	27.89	27.22	26.79

The preparative decomposition of magnesite may be carried out even at 500°C when a suitable atmosphere, e.g., air containing  $H_2O$ and  $NH_3$  at a total partial pressure of 40 mm., is provided [19]. The gaseous atmosphere over the preparation also affects the particle size of the nascent solid phase. Thus, given identical decomposition temperatures, the particle size of MgO crystals formed from magnesite will decrease with decrease of the  $CO_2$ pressure during the decomposition [10].

#### FORMATES AND OXALATES

The nature of the thermal decomposition of formates and oxalates is not uniformly the same. Depending on the basicity of the metal and the degree to which it approaches the noble metals, one may obtain carbonates, oxides or the metals themselves, provided secondary oxidation reactions are prevented (the production of metals is due to reduction by the organic anions or their decomposition products). According to presently available data, formates and oxalates behave in this respect in practically identical fashion. As far as is known, the composition of products obtained on vacuum decomposition is identical to that of the primary products of decomposition in air. The following data are available on some of the divalent metals:

Formate or oxalate of	Decomposition products in vacuum and primary products of decom- position in air	Terminal products of decomposition in air		
Mg*	MgO	MgO		
Mn	MnO	Higher oxides		
Fe	FeO	$\alpha$ - and $\gamma$ - Fe <sub>2</sub> O <sub>3</sub>		
Co	Со	Higher oxides		
Ni	Ni	$NiO_1 + X$		
Cu	Cu	CuO		
Zn	ZnO	ZnO		

\*At the mild decomposition temperatures assumed in this table, the salts of the heavier alkaline earths always produce carbonates.

Under certain decomposition conditions, Cd and Pb (II) oxalates give a mixture of oxide and metal [6, 11, 15]. In contrast to the corresponding formates, the oxalates of the metals mentioned here are rather insoluble in water; in general, they are readily obtained by precipitation, which usually gives a fine powder well suited as a starting material for decomposition. Provided the solvent is cautiously removed, hydrated or solvated compounds also yield suitable starting materials, and decomposition of these gives products with high activities.

Ultramicroscopic and x-ray studies show that oxides obtained from carbonates and oxalates at low decomposition temperatures are pseudomorphous with the crystals of the starting material [10, 12]. Since the molar volume of the new substance is as a rule considerably smaller than that of the starting material, the particles of the product are usually very porous, that is, provided the reduced decomposition temperature  $T/T_m < 1/3$  (compare p. 1611). If the decomposition temperature is higher, aggregate crystallization can be expected to an increasing degree [19], except when the starting substance (which must in this case be a uniform fine powder) is heated for such a short time that only the desired

reaction occurs and no time is left for the material to undergo aggregation [20]. Fast cooling from the decomposition temperature may help in this case.

The following are literature references for the preparation of active oxides by thermal decomposition of suitable compounds:

Droparation	Formation from							
Preparation	Nitrate	Carbonate	Formate	Oxalate				
BeO		See p. 893						
MgO		[3, 7, 22]	[25]	[16, 23]				
Mn oxides	[9]; see p. 1458	See p. 1456	[25]	[23]; see p. 1456				
Fe oxides			[5]	[5, 23]; see p. 1497				
Co oxides		See p. 1519	[25]	[23]				
Ni oxides	[14, 26]; see p. 1548	[14, 24]; see p. 1548	[25]	[1, 23]				
CuO	[24]; see p. 1012		[25]					
ZnO	[18]	[18, 27]	[25]	[18, 23]				
CdO			[25]	[11, 15]				
ThO2	See p. 1221			[2, 13]; see p. 1221				

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# Lead (IV) Oxide

 $\begin{array}{rl} Pb(CH_{3}COO)_{4} + 2 H_{2}O &= PbO_{2} + 4 CH_{3}COOH \\ 443.4 & 239.2 \end{array}$ 

The instructions given on p. 767 are used to prepare 50 g. of  $Pb(CH_3COO)_4$ . The material is then crushed and triturated with 460 ml. of water in centrifuge tubes until all tetraacetate is converted to brown  $PbO_2$ . The fine suspension is then centrifuged. The deposit is stirred four times with water (500 ml. each time); the suspension is centrifuged each time before decanting the liquid. The last wash water should not be acid. The PbO<sub>2</sub> is then suspended in 50 ml. of water, filtered with suction, and washed on the filter with 50 ml. of water. Then the material is washed on the funnel with four 25-ml. portions of acetone to displace the water. The acetone is then displaced by washing with four 25-ml. portions of ether. This imparts the final color to the material. The product is immediately placed in a vacuum desiccator [5].

PROPERTIES:

Very fine, dense powder with a light brown, coffeelike color; reacts with dil. HCl to give chlorine. Its physical state, characterized by small particle size, lattice defects and occlusions of admixtures which prove to be amorphous and on x-ray analysis are recognizable under an ultramicroscope [6], causes an extraordinary high chemical activity. Especially useful in dehydrogenation of aromatic dihydroxy compounds to the corresponding quinones. Pure preparations age with release of  $O_2$ , losing their activity (5% in 15 hours, 8-11% in 7 days).

#### GENERAL:

Spontaneous dehydration of hydroxides is occasionally also observed in precipitations with alkali hydroxide solutions. The method is especially useful in preparation of CuO [2, 7], ZnO [3] and  $Ag_2O$  [4]. The products thus obtained are usually fine powders and quite frequently contain occluded admixtures which prove amorphous on x-ray analysis. The rate of dehydration depends on the particle size of the precipitated hydroxide and on the possible formation of intermediate basic salts. The latter may inhibit the dehydration to a considerable extent. An especially useful starting material for preparation of CuO is Cu(NO<sub>3</sub>)<sub>2</sub> [1].

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# Colloidal Suspensions of Oxides in Gases (Smokes)

Oxides of especially small particle size (smokes) may be obtained under appropriate reaction conditions. Such reactions are nearly always described in conjunction with special investigations where only very small amounts of compound are necessary. Only occasional literature references to work on a preparative scale are available. However, these may be supplemented by data and descriptions from the patent literature, since oxide smokes are prepared on an industrial scale. The following methods are the most useful for generation of oxide smokes:

1) Burning of the metal (e.g., Mg, Zn, Cd and so forth); however, this procedure frequently converts only part of the oxide to the desired colloidal (smoke) dispersion.

2) Oxidation of a metal volatilized in an electric arc. This simple method is frequently used to demonstrate the present phenomenon. The procedure consists of striking an arc between two electrodes made of the desired metal. However, it is only rarely useful for preparative purposes, since the quantity of product is very small due to the high thermal conductivity of the electrodes and the frequent shifts in the arc focus.

A much higher efficiency is reached in the apparatus developed by V. Kohlschütter [5] (see Fig. 340), comprising an arc furnace with vertical electrodes. The bottom electrode has an indentation for the metal to be volatilized. The arc is struck between the metal and the top (movable) electrode. Usually the vaporization proceeds very smoothly if the metal is the anode. The optimum conditions for the operation of the arc and for removal of the vapors from the electrodes vary somewhat from metal to metal.



Fig. 340. Preparation of active metal oxides by oxidation of metal vapor. a funnel for addition of metal; b observation port; c side port; d circular nozzle for air intake; e first chamber with lateral observation ports (these are not shown); f illuminating device; g glass tubes (the remaining parts of the apparatus are made from sheet iron); h carbon electrodes; i flow meter activated by differential pressure; n precipitation cell; s movable carbon electrode.

3) Thermal or photochemical decomposition of volatilized metal compounds in the presence of oxygen. Metal compounds that

are usable in such reactions are carbonyls and certain metal hydrides or organometallic compounds, for example, alkylated metals. To avoid explosions, the reaction must be carried out at very low pressures or the partial pressure of the reagents must be reduced by appropriate dilution with an inert gas. The reactions may be carried out either in closed systems [4, 10] or in a gas stream [11]. The continuous apparatus is shown in Fig. 338.

The decomposition of  $Fe(CO)_5$  has been investigated very thoroughly. It was found that it oxidizes according to the following equation:

4 Fe(CO)<sub>3</sub>  $\xrightarrow{(O_2)}$  2 Fe<sub>2</sub>O<sub>3</sub> + xCO<sub>2</sub> + yCO 738.6 319.4

Low  $O_2$  concentrations in the starting mixture, high temperatures and short reaction times give thoroughly crystalline  $\gamma$  -Fe<sub>2</sub>O<sub>3</sub>; this material gives sharp powder patterns. Lower decomposition temperatures and higher oxygen partial pressures produce smaller smoke particles; however, the particles show powder patterns which are less distinct [11]. Similar investigations have also been carried out on the decomposition of Pb(CH<sub>3</sub>)<sub>4</sub> [6].

4) Decomposition of halide vapors. Readily vaporized halides, especially the chlorides of elements such as Al, Si, Ti, Zr, Sn and so forth, may also be converted to oxides in the gas phase. This may be done in either of the following ways:

a) Saturation of a stream of an inert gas with the chloride, followed by reaction with steam. External heating is required in this case.

b) Combustion of the chloride together with hydrogen or with a combustible gas which contains bound hydrogen, e.g.:

 $2 \operatorname{SiCl}_{4} + 2 \operatorname{CH}_{4} \xrightarrow{(O_{2})} 2 \operatorname{SiO}_{2} + 8 \operatorname{HCl} + x \operatorname{CO}_{2} + y \operatorname{CO}_{339.8} \qquad 32.1 \qquad 120.1$ 

The second gas may be preheated or the combustion mixture may be diluted with an inert gas (to avoid excessive temperatures), depending on circumstances.

Under suitable reaction conditions this procedure yields extremely fine oxide powders which are used industrially as active, white fillers [2, 7].

Some difficulties in the preparation of oxide smokes arise in the separation and collection of the smoke particles. The apparatus of Kohlschütter solves the problem partially by using a dust collector and a Cottrell precipitator. However, losses are still heavy due to elutriation of the smallest particles. The following are literature references for the preparation of oxide smokes from metals or carbonyls:

Oxide	MgO	Al <sub>2</sub> O <sub>3</sub>	Cr <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	NiO	CuO	ZnO	CdO	SnO₂	PbO	Bi <sub>2</sub> O <sub>3</sub>
from metal	[1]	[5]	[5]	[1,9]	[1,5]	[5]	[1,5]	[1,3,5]	[5]	[5]	[5]
from carbonyl				[1 <b>,</b> 4, 8, 10, 11]							

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# Copper-Chromium Oxide

 $\begin{array}{rl} (\mathrm{NH_4})_2\mathrm{Cr_2O_7} + 4\,\mathrm{NH_3} + 3\,\mathrm{H_2O} + 2\,\mathrm{Cu}(\mathrm{NO_3})_2 &= 2\,\mathrm{Cu}(\mathrm{OH})\mathrm{NH_4CrO_4} \\ & & (3\,\mathrm{H_2O}) \\ & & 252.1 & 68.1 & 483.3 & 429.3 \\ & & + 4\,\mathrm{NH_4NO_3} \\ & 2\,\mathrm{Cu}(\mathrm{OH})\mathrm{NH_4CrO_4} &= \mathrm{CuO} + \mathrm{CuCr_2O_4} + 5\,\mathrm{H_2O} + \mathrm{N_2} \\ & & 429.3 & 79.6 & 231.6 \end{array}$ 

A solution of 126 g. of  $(NH_4)_2Cr_2O_7$  A.R. in 600 ml. of water is prepared, and 150 ml. of 28% aqueous ammonia is added to it. This solution is poured in a thin stream into a warm (50-60°C) solution of 242 g. of  $Cu(NO_3)_2 \cdot 3 H_2O$  in 800 ml. of water while stirring manually. The reddish-brown precipitate of  $Cu(OH)NH_4CrO_4$  [3] is stirred for an additional few minutes and then filtered on a Büchner funnel. The moisture is pressed out and the filter cake dried in an oven at 110°C. The mass is then broken up into coarse pieces and heated in a covered nickel or porcelain dish in a muffle furnace  $(350-450\,^{\circ}\text{C}$ , one hour). The product  $(\text{CuO} + \text{CuCr}_2\text{O}_4)$ [10, 11] is ground in a mortar and suspended three times in 10% acetic acid (1.2 liters each time); the product is settled and the liquid decanted after each washing. This removes the CuO. The residue is washed four times with water in the same manner, filtered with suction, dried at 110°C and ground. After the last washing the precipitate sometimes settles only with difficulty due to partial peptization [8].

After this treatment, the catalyst may still contain an excess of CuO; the latter may convert during use to  $Cu_2O$  [9], which decreases the catalytic activity. The deactivation of the catalyst is much less likely if 24 g. of the  $Cu(NO_3)_2 \cdot 3 H_2O$  in the initial charge is replaced by 26 g. of  $Ba(NO_3)_2$ . The Ba appears in the product catalyst as  $BaCrO_4$  [11]. Equivalent quantities of Mg, Ca, Mn (II) or Zn nitrates may be used instead of the  $Ba(NO_3)_2$ .

#### PROPERTIES:

Fine black powder; completely stable in atmospheric oxygen and moisture. Active catalyst for the hydrogenation of organic hetero compounds containing multiple bonds [1, 5].

#### GENERAL:

Even the simple ammonium chromates can undergo thermal decomposition [4, 6, 7, 12] which yields very active Cr (III) oxides on decomposition in the air. Partial formation of higher chromium oxides is observed at reduced pressures [6]. Since these decomposition reactions release a considerable quantity of heat, they must be carried out in thin layers (flat dishes); the mildest conditions are obtained if only one spot in the dish is heated at a time and that spot is pushed onto a cold surface as soon as the material starts to react. The thermal decomposition of  $NH_4MnO_4$  in air leads to explosions; vacuum decomposition yields, apart from  $NH_4NO_3$ , manganese oxides [Mn (III)-Mn (IV)], which are pseudomorphous to the starting crystals [2].

The above method is also useful for the production of other mixed oxides where chromium oxide is a constituent, e.g., Mn-Cr oxide, Zn-Cr oxide [13].

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### Hopkalite (Hopcalite)

#### HOPKALITE I

Hopkalite I is a mixture of 50% MnO, 30% CuO, 15%  $Co_2O_3$  and 5%  $Ag_2O$  [9]. Its constituents are prepared as follows:

a) Manganese dioxide [2, 5]

 $2 \text{ KMnO}_4 + 5 \text{ H}_2 \text{SO}_4 = \text{ K}_2 \text{SO}_4 + 2 \text{ Mn}(\text{SO}_4)_2 + 5 \text{ H}_2 \text{O} + 1^1/_2 \text{ O}_2$ 316.1

$$2 \text{ Mn}(\text{SO}_4)_2 \ + \ 4 \text{ H}_2\text{O} \ = \ 2 \text{ MnO}_2 \ + \ 4 \text{ H}_2\text{SO}_4$$

$$173.9$$

Cold 75% sulfuric acid (650 g.) is poured over 100 g. of fine  $KMnO_4$  powder; the mixture is left standing for several days. During this time, the initially separating  $HMnO_4$  decomposes with evolution of oxygen, leaving a dark yellow Mn (IV) solution. This is added to a large excess of water; a very fine powder of hydrated  $MnO_2$  separates out. This powder is washed several times by decantation with water. The washing is continued on a funnel until the filtrate is free of sulfate.

Alternate method: Reduction of permanganate by Mn (II) salts [1, 4, 13].

b) Copper oxide [9] (see also [6]):

A solution of 103.5 g. of  $CuSO_4 \cdot 5 H_2O$  in 300 ml. of water is poured with efficient stirring into 430 ml. of 2N NaOH preheated to about 80 °C. The mixture is stirred for a few minutes; during this time,  $2N H_2SO_4$  is added until the solution, which contains the CuO in suspension, becomes neutral (about 15 ml. of the acid is required). The mixture is allowed to settle and washed in the same way as the  $MnO_2$  (see above).

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c) Cobalt (III) oxide [8]
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A solution of 56.2 g. of  $CoSO_4 \cdot 7 H_2O$  in 200 ml. of water is mixed at room temperature with a slight excess of strongly alkaline hypochlorite solution. As soon as gas evolution comes to a virtual stop, the solution is decanted from the immediately appearing dense black precipitate, which catalyzes the decomposition of excess hypochlorite. The washing procedure is the same as for MnO<sub>2</sub>.

The oxides prepared as in (a), (b) and (c) are suspended in about 1.5 liters of water. This suspension is then mixed with a solution of 8.07 g. of AgNO<sub>3</sub> in the minimum amount of water, and the Ag<sub>2</sub>O is precipitated by addition of 23.8 ml. of 2N NaOH (intensive stirring). Further treatment is given under Hopkalite II.

# HOPKALITE II

Hopkalite II consists of 60% MnO<sub>2</sub> and 40% CuO (the MnO<sub>2</sub>:CuO molar ratio is 1.375). The catalyst may be prepared by mixing the separately prepared components [10] or by mixed precipitation ([13]; see also [11]).

a)	$MnSO_4 + 2$	$MnSO_4 + 2 NaOH + \frac{1}{2}O_2 = MnO_2 + Na_2SO_4 + H_2O$					
	$(4 H_2 O)$ 223.1	80.0	86.9				
b)	CuSO <sub>4</sub>	+ 2 NaOH	$= \mathrm{CuO} + \mathrm{Na_2SO_4} + \mathrm{H_2O}$				
	$(5 H_2 O)$ 249.1	80.0	79.6				

The reactor is a three-liter Erlenmeyer flask, in which 155 g. of  $MnSO_4 \cdot 4 H_2O$  and 125 g. of  $CuSO_4 \cdot 5 H_2O$  are dissolved in 1.5 liters of hot (70-80°C) water. A fast stream of air is then bubbled through the hot solution and 400 ml. of 25% NaOH is added (vigorous shaking) from a dropping funnel. The passage of air is continued for another 10 minutes. The precipitate is then washed several times by decantation with hot water. Then washing is continued on a filter until the wash liquor is neutral. The precipitate is suction-dried (the water should be removed as completely as possible by repeated pressing). After the by-product fine powder has been sieved out, the product is granulated by heating for 3 hours at 200  $^\circ\text{C}.$ 

PROPERTIES:

Brownish-black granules, which must be stored under anhydrous conditions. Catalyzes the combustion of CO at room temperature; used in gas-mask cartridges.

GENERAL:

The methods of preparation described above consist of precipitation immediately followed by a reaction of the precipitated product (dehydration, oxidation). The oxidation is a topochemical reaction yielding products with defect structures. The full activity of such a multiple-compound catalyst can frequently be developed only after an aging process called forming. To strengthen the structure of the catalyst granules (which at the same time increases the accessibility of their internal surfaces to gases), rather large amounts (up to 50% and more) of kieselgur may be added to the catalyst (as, for instance, in the Fischer-Tropsch catalysts).

Some other oxidation catalysts based on metal oxides have been described in the literature:  $Ag_2O-Cu_2O$  [7];  $Ag_2O-Cr_2O_3$  [12] and  $MnO_2-CuO-Co_xO_y$  [3].

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# SECTION 2

Hydroxo Salts

**R. SCHOLDER** 

### General

"Hydroxo salt" is the term used for a group of complexes where the central atom of the complex anion is a metal to which hydroxyl ions are bound as ligands. The number of these ions depends on the normal coordination number of the metal. The cation of a hydroxo salt is usually an alkali metal, particularly sodium, or the alkaline earth metals barium, strontium and, in some cases, calcium. Heavy-metal salts can be prepared from a few hydroxo anions via a double decomposition reaction.

Hydroxo salts correspond closely to the well-known halo salts in their formula type and structure. Accordingly, mixed halohydroxo salts of a number of metals can be prepared. At elevated temperature, hydroxo salts can be converted into oxo salts, provided it is feasible to prepare the latter from a metal hydroxide (the central atom) and an alkali or alkaline earth metal hydroxide. In numerous systems, however, such oxometalates can be obtained only from oxides. In some cases, oxohydroxo salts are also formed as intermediates.

Complex hydroxometal anions are formed in solution via the following equilibrium reaction:

 $[M(OH)_{x} \rightleftharpoons M^{x(+)} + x OH^{-}]$  $M^{x(+)} + y OH^{-} \rightleftharpoons [M(OH)_{y}]^{(y-x)} (-)$ 

The reagents are strong bases and poorly soluble metal hydroxides. In the above equations, the metal hydroxides function as "acids," in agreement with the modern theory of their amphoteric behavior. As far as is presently known, the following metals (arranged in order of increasing valence) form hydroxo salts:

M(II): Be, Mg, Sn, Pb, Mn, Fe, Co, Ni, Cu, Zn, Cd
M(III): Al, Ga, In, Bi, Cr, Mn, Fe
M(IV): Sn, Pb, Pt
M(V): Sb

The equilibrium distribution is of controlling importance in the preparation of crystalline alkali hydroxometalates. Most hydroxo salts, alkali salts in particular, rapidly decompose into their components in the presence of  $H_2O$  or dilute alkali hydroxide. Only the hexahydroxo salts of Sn (IV), Pt (IV) and Sb (V) dissolve in  $H_2O$  at room temperature without decomposition, whereas the other alkali hydroxometalates are stable only when they constitute the solid phase in the presence of (usually very concentrated) alkali hydroxide solution. The alkaline earth hydroxometalates are relatively sparingly soluble and hence are stable in dilute hydroxide; the compound  $Ca[Zn(OH)_3]_2 \cdot 2 H_2O$  is stable even in water. The equilibrium shifts toward the reagents with temperature.

Metal oxide-sodium oxide-water systems at constant temperature exhibit the same general behavior regardless of the particular compounds involved (see Fig. 341). Thus, with increasing alkali hydroxide concentration, the solubility at first increases to a maximum and then decreases sharply. The rising branch of the curve corresponds to solid  $M(OH)_m$  or  $MO_n$  while the decreasing branch corresponds to sodium hydroxometalate, whose solubility is sharply reduced as the NaOH concentration increases.



Fig. 341. Solubility of the system metal oxide-sodium oxide-water as a function of sodium hydroxide concentration.

The experimental fact that the sodium hydroxide concentration at the solubility maximum is usually over 30% indicates that highly concentrated (usually 45-50%) NaOH is required for the preparation of most alkali hydroxo salts. This is also necessary to ensure good yields. However, most alkaline earth hydroxometalates can be prepared from more dilute NaOH solutions.

Spreading the microcrystalline solid phase on a clay dish does not afford complete separation from the residual mother liquor. It has recently been established that alkali hydroxometalates can be separated from the mother liquor much more efficiently by brief shaking with pure isoamyl alcohol, and the alkaline earth salts by treatment with anhydrous methanol, possibly containing a small amount of NaOH. Several types of hydroxo salts of some di- and trivalent metals can be prepared just as in the case of halo salts. These differ in the number of coordinated  $OH^-$  ions. The number of  $OH^-$  ligands depends on the concentration and temperature of the alkali hydroxide.

Among the less stable alkali hydroxometalates, it is often only the Na salt, but not the K salt, that can be prepared. This is due to the unusually high solubility of the corresponding K salts. In fact, the precipitation of the latter even from highly concentrated hydroxide solutions is often impossible without simultaneous crystallization of KOH. The rather sparingly soluble Ba and Sr salts can be precipitated from the metal hydroxide or oxide solutions by addition of  $Ba^{2+}$  or  $Sr^{2+}$ . The alkaline earth salts can be obtained more conveniently by simultaneous dropwise addition, in proper ratios, of concentrated solutions of the perchlorates of the central metal atom and of the alkaline earth metal to hot, moderately concentrated sodium hydroxide (sodium perchlorate is much more readily soluble in strong sodium hydroxide than is NaCl). The free hydroxo acids, which should exist as well-defined higher hydrates of the metal oxides, are not known, the exception being hexahydroxoplatinic (IV) acid,  $H_2[Pt(OH)_6]$ .

In some cases the fact that the hydroxometalates are chemical complexes is indicated by the color of the salts and of their solutions. The proof of structure is based on their thermal dehydration curve, their ability to form mixed halo-hydroxo salts, data on isomorphic relations, and some powder pattern studies.

# Handling of Concentrated Alkali Hydroxides

### STARTING MATERIALS

Very pure or reagent grade (97-98%) NaOH pellets and a similar grade of potassium hydroxide (containing an average of 85% of KOH, the remainder being  $H_2O$ ) are used.

### CONTAINER MATERIALS

Chemical glassware is sufficiently resistant to concentrated alkali hydroxide solutions at room temperature that it can be used without adversely affecting the purity of the products. However, hot, concentrated alkali hydroxide solutions attack any glass so strongly that the latter can be used at high temperatures only for short periods of time, if at all. Such experiments must therefore be run in refined silver containers, which resist even concentrated boiling alkali hydroxide solutions. While pure nickel containers are also suitable, they are not cheaper than silver.

### FILTERS

Fritted Pyrex glass of medium and high porosity is suitable. However, its life may be limited in repeated use for filtering hot alkali hydroxide solutions. When filtering very hot and concentrated alkali hydroxide solutions, the glass suction funnel should be wrapped with a strong cloth so as to reduce as much as possible the ever-present danger of sudden breakage (this danger is real even with infrequently used filters). To prevent solidification of solutions containing more than 50% NaOH during filtration, the glass suction funnel is surrounded with a sheet metal jacket containing hot glycerol. When such solutions are boiled in a flask, the rubber stopper must be protected with an asbestos liner.

Even though filter plates made of certain plastics resist hot concentrated alkali hydroxide solutions, we have not yet tested them sufficiently to recommend them for laboratory use.

### CARBONATE-FREE SODIUM HYDROXIDE

A 50% sodium hydroxide solution is prepared from the calculated amount of commercial NaOH in a silver flask. To prevent the occasional nuisance of the NaOH sticking to the bottom, the flask is vigorously shaken; solution is promoted by the strong, spontaneous heat evolution. However, external heating should be avoided because of the danger that the hot caustic solution will bump and spill out of the flask, a danger not obviated by the presence of a reflux condenser on the flask. The 50% hydroxide solution is allowed to cool slowly and, if possible, to stand at room temperature for 2-3 days. The precipitated Na<sub>2</sub>CO<sub>3</sub> is then filtered off on glass frit of small pore size (rigorous exclusion of air). The completely clear filtrate is virtually carbonate-free. To avoid waiting for precipitation of the carbonate, one can add 1-2 g. of Ba(OH)<sub>2</sub> per 100 ml. of hot, 50% sodium hydroxide; the mixture may then be filtered immediately after cooling to room temperature.

Sodium hydroxide solutions of lower concentration can be prepared by dilution of the 50% solution with boiled H<sub>2</sub>O. Removal of Na<sub>2</sub>CO<sub>3</sub> from a solution containing more than 50% NaOH must be carried out at 40-60°C to prevent crystallization of NaOH. Such solutions may also be prepared by distilling the calculated amount of water from carbonate-free 50% sodium hydroxide into a graduated cylinder. For obvious reasons, CO<sub>2</sub> must be rigorously excluded. All equipment, including the reflux condenser and the glass suction filter, is protected with Peligot tubes (containing 50% KOH) held in place by rubber stoppers. Handling alkali hydroxide solutions is often facilitated by a knowledge of their boiling points. For this reason, the boiling points of 20-70% NaOH and KOH solutions, taken from Gerlach [2], are tabulated below.

	25	42.8	53 <b>,8</b>	66.7	81,8	100	122,2	150	233.3	g. NaOH (KOH)/ 100 g. H 2O
	20	30	35	40	45	50	55	60	70	% NaOH (KOH)
NaOH	108	116	121,5	128	134.5	142,5	150,5	160	180,5	B.p., °C
кон	106	113	118	124,5	133	145	160,5	177,5	228	B.p., °C

### SAFETY RULES

The destructive action of caustic alkali solutions begins immediately. Therefore, the eyes must always be protected by goggles which fit tightly on all sides. Any caustic solution under the fingernails should immediately be washed off with a large amount of water followed by dilute acetic acid. Silver being a much better heat conductor than glass, one should remember that silver equipment will get hot much more rapidly than glass.

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### Sodium Hydroxozincates

The system  $ZnO-Na_2O-H_2O$  contains the following four solid hydroxozincates, whose existence depends on the NaOH concentration:

 $Na[Zn(OH)_3] \cdot 3 H_2O; Na[Zn(OH)_3]; Na_2[Zn(OH)_4] \cdot 2 H_2O; Na_2[Zn(OH)_4].$ 

SODIUM TRIHY DROXOZINCATE, Na [Zn(OH)3]

 $ZnO + NaOH + H_2O = Na[Zn(OH)_3]$ 81.4 40.0 18.0 139.4
## PREPARATION OF CRUDE MATERIAL

A hot solution of 185 g. of NaOH in 100 ml. of H<sub>2</sub>O is prepared, and ZnO (105 g.) is added. The mixture is refluxed 0.5 hours and cooled to 100°C. Water (85 ml.) is then gradually added through the condenser (use a funnel). The solution is filtered hot to remove residual ZnO and the filtrate is immediately cooled to about 15°C. If crystallization does not set in within a day, some NaOH pellets are dissolved in a few milliliters of the boiling zincate solution, the solution is cooled in a freezing mixture, and the resulting tetrahydroxozincate, which precipitates readily at this higher NaOH concentration, is used to seed an additional 5ml. portion of the original zincate solution (moderate cooling). Rubbing with a glass rod initiates crystallization of the trihydroxozincate. The entire zincate solution is then seeded with this material. The precipitate obtained is filtered off after a few hours, washed with 50% sodium hydroxide, and dried on a clay plate in an empty desiccator. Yield: 50-60 g.

#### B. PREPARATION OF THE PURE COMPOUND

A solution (prepared at the boil) of 60 g. of ZnO and 250 ml. of pure, 51% sodium hydroxide is filtered at about 40°C, cooled to 15°C, and seeded with zincate prepared as described under (A). After 12 hours the mixture is filtered and worked up as under (A). Yield 40 g.

The moist product (10 g.) is shaken for two hours with 150 ml. of alkaline methanol solution (100 ml. of  $CH_3OH$  plus 15 g. of NaOH), filtered, washed first with the same methanol solution and then repeatedly with acetone, and dried over silica gel. This method removes the last traces of NaOH, and the analysis shows the calculated percentages of ZnO, Na<sub>2</sub>O and H<sub>2</sub>O.

#### PROPERTIES:

Colorless, microcrystalline powder (small rods). Decomposes immediately in water; decomposes after a few hours in 10% methanolic NaOH; stable in 15% methanolic NaOH ( $18^{\circ}$ C). When prepared by method (A), contains about 0.1 moles of NaOH per mole of zincate.

#### SODIUM TETRAHYDROXOZINCATE, Na<sub>2</sub> [Zn(OH)<sub>4</sub>]

A carbonate-free, clear solution of 195 g. of NaOH in 140 ml. of  $H_2O$  is prepared, and ZnO (56 g.) is dissolved in it at the boil; the mixture is filtered at 90°C. The crystals that separate out after a few hours are washed with 50% NaOH and spread in as thin a

layer as possible on a clay dish. They are dried in an empty desiccator. Yield: about 100 g.

PROPERTIES:

Formula weight 179.42. Microcrystalline, thin platelets. The NaOH traces (about 0.2 moles per mole of zincate) cannot be removed.

REFERENCES:

R. Scholder and H. Weber. Z. anorg. allg. Chem. <u>215</u>, 355 (1933);
 R. Scholder and G. Hendrich. Ibid. <u>241</u>, 76 (1939);
 R. Scholder and K. Osterloh. Unpublished data.

#### Sodium Tetrahydroxomagnesate

# Na<sub>2</sub>[Mg(OH)<sub>4</sub>]

 $\begin{array}{rll} Mg(OH)_2 \ + \ 2 \ NaOH \ = \ Na_2[Mg(OH)_4] \\ 58.3 & 80.0 & 138.3 \end{array}$ 

An approximately 65% NaOH solution is prepared in a silver flask by distilling 180 ml. of H<sub>2</sub>O from 500 ml. of 50% sodium hydroxide. The solution is cooled to about 100°C and 6 g. of Mg(OH)<sub>2</sub>, prepared by slaking MgO (calcined at 500°C) with hot  $H_2O_1$ , is added to it. The mixture is agitated with a silver stirrer and refluxed for 20 hours at 100°C in the absence of CO<sub>2</sub>. Without interrupting the heating, the flask contents are transferred by suction (use silver tubing interconnected with polyethylene sleeves) onto a glycerol-heated, medium-pore-size fritted-glass filter, which is maintained at 100°C. The filter cake is dried by suction and immediately spread on a clay dish heated to 100°C. The dish is then kept for about five hours in a vacuum desiccator heated to 100°C, to promote absorption of the surface sodium hydroxide by the clay plate. This procedure yields 8-10 g. of relatively dry sodium hydroxomagnesate which is strongly contaminated with NaOH. To remove the NaOH, 3 g. of the crude product is pulverized in the absence of CO<sub>2</sub> and H<sub>2</sub>O, and then shaken for 30-45 min. with freshly distilled isoamyl alcohol (b.p. 127-129°C). The mixture is suction-filtered (again in the absence of CO<sub>2</sub>) through a medium-pore-size fritted-glass filter, and rinsed with 50 ml. each of isoamyl alcohol and ether. The product is then dried for a few hours over silica gel while simultaneously removing the ether in vacuum.

PROPERTIES:

Microcrystalline hexagonal platelets. Yields crystalline  $Mg(OH)_2$  (brucite) on treatment with  $H_2O$ . Decomposed by strongly alkaline methanol or ethanol even below 0°C; isoamyl alcohol gradually splits off NaOH, but only on prolonged reaction.

**REFERENCE**:

R. Scholder and C. Keller. Unpublished data.

### Sodium Tetrahydroxocuprate (II)

#### Na<sub>2</sub>[Cu(OH)<sub>4</sub>]

 $CuO + H_2O + 2NaOH = Na_2[Cu(OH)_4]$ 79.5 18.0 80.0 177.6

#### A. CRUDE MATERIAL

Very pure CuO (15 g.) is dissolved in a clear, carbonate-free solution of 500 g. of NaOH in 330 ml. of  $H_2O$  (brief refluxing). The dark-blue solution is cooled to 110°C and carefully diluted by adding 140 ml. of  $H_2O$  through the reflux condenser (use a funnel). The small quantity of unreacted CuO is then filtered off, collecting the filtrate in a preheated Erlenmeyer flask of refined silver. The Erlenmeyer flask is protected by a Peligot tube (filled with 50% KOH) and kept in an electric drying oven for six days at 75°C to allow the filtrate to crystallize. The mixture is then filtered; the crystals are washed with some 50% and 45% sodium hydroxide (once each) at room temperature and dried on a clay plate over  $H_2SO_4$ . Yield: 13 g.

#### **B. PURIFICATION**

The considerable amount of NaOH still present in the product is removed immediately following the washing with the 50% NaOH. Thus the dark-blue crystals are shaken for one hour with 150 ml. of 40% NaOH at room temperature and filtered.

The crystals are then shaken for one minute with the following solutions (in the order given):  $150 \text{ ml. CH}_3\text{OH} + 22.5 \text{ g. NaOH}$  (18°C);  $150 \text{ ml. CH}_3\text{OH} + 15 \text{ g. NaOH}$  (0°C);  $150 \text{ ml. CH}_3\text{OH} + 1.5 \text{ g. NaOH}$  (-10°C).

After decantation, the solid is finally digested twice with pure methanol ( $-10^{\circ}$ C), filtered and washed with methanol at  $-15^{\circ}$ C. The crystals are placed on a clay plate and dried over silica gel in a minimum size desiccator.

When the cuprate solution remaining after filtering off the CuO is quickly cooled to room temperature in a freezing mixture, the salt precipitates as very thin, light-blue platelets. These, however, cannot be completely freed of the excess NaOH. Yield: 20 g.

PROPERTIES:

Firm, dark-blue crystals. The very pure salt obtained by method (B) is extremely sensitive to moisture and rapidly turns dark brown on exposure to air.

REFERENCES:

R. Scholder, R. Felsenstein and A. Apel. Z. anorg. allg. Chem. <u>216</u>, 138 (1934); R. Scholder and K. Osterloh. Unpublished data.

#### Barium Hexahydroxocuprate (II)

#### Ba<sub>2</sub>[Cu(OH)<sub>6</sub>]

 $\frac{\text{Na}_{2}[\text{Cu}(\text{OH})_{4}] + 2 \text{Ba}(\text{OH})_{2}}{177.6} = \frac{\text{Ba}_{2}[\text{Cu}(\text{OH})_{6}] + 2 \text{Na}\text{OH}}{440.3} = \frac{128}{80.0}$ 

A solution of 10 g. of  $CuBr_2$  in 25 ml. of  $H_2O$  is added to 200 ml. of carbonate-free 50% sodium hydroxide at +5°C. The resulting mixture is heated to 70°C (water bath) and the small amount of CuO filtered off. The filtrate is refluxed 130°C and a hot solution of 30 g. of Ba(OH)<sub>2</sub>.8 H<sub>2</sub>O in 50 ml. of H<sub>2</sub>O is added to it through a fluted filter (shaking). The salt that separates is immediately filtered off, cooled to 0°C in an Erlenmeyer flask, shaken for 5 min. with 100 ml. of methanol at -10°C, and filtered off. It is washed with methanol at 0°C and then thoroughly with acetone and anhydrous ether. The residual ether is removed by prolonged vacuum treatment in a desiccator. The product is completely pure. Yield: 13 g.

PROPERTIES:

Light-blue powder (rhombic crystal aggregates). Decomposed by  $H_2O$ .

REFERENCES:

R. Scholder, R. Felsenstein and A. Apel. Z. anorg. allg. Chem. <u>216</u>, 138 (1934); R. Scholder and V. Voelskow. Unpublished data.

# Sodium Tetrahydroxoferrate (II)

#### $Na_2[Fe(OH)_4]$

The reactor is a round-bottom, refined-silver flask, carrying a rubber stopper holding a reflux condenser protected by a Peligot tube (filled with an alkaline pyrogallol solution) and a silver tube serving as inlet for pure nitrogen. The flask is charged with 8 g. of reduced iron and 350 ml. of a 50% solution of pure NaOH. The air is displaced with N<sub>2</sub> and the mixture is refluxed for 2.5 hours in a steady N<sub>2</sub> stream. The blue solution is cooled to 120°C and suction-filtered (in the absence of air) through a glass frit covered with a layer of reduced iron. The filtrate is collected in a Pyrex suction flask containing 100 ml. of 50% sodium hydroxide, through which a nitrogen stream may be passed. The filtrate is allowed to cool for about 12 hours under N<sub>2</sub>; the gray-green precipitate is then filtered off under N<sub>2</sub>, washed with 50% sodium hydroxide, and dried on a clay dish in a nitrogen-filled desiccator. Yield: 4 g.

PROPERTIES:

Gray-green microcrystalline powder (polyhedral crystals); very sensitive to moisture and O<sub>2</sub>. Besides the polyhedra, microscopic examination also reveals colorless platelets with oblique sides of Na<sub>4</sub>[Fe(OH)<sub>7</sub>]  $\cdot$  2 H<sub>2</sub>O (see p. 1689).

REFERENCE:

R. Scholder. Angew. Chem. 49, 255 (1936).

#### Strontium Hexahydroxonickelate (II)

### Sr<sub>2</sub>[Ni(OH)<sub>6</sub>]

 $\begin{array}{rrrr} \text{Ni}(\text{ClO}_{4})_{2} &+& 2\,\text{Sr}(\text{ClO}_{4})_{2} &+& 6\,\text{NaOH} &=& \text{Sr}_{2}[\text{Ni}(\text{OH})_{6}] &+& 6\,\text{NaClO}_{4} \\ && 257.6 & 573.1 & 240.0 & 336.0 & 734.7 \end{array}$ 

A mixture of 250 g. of NaOH and about 8 g. of  $Sr(OH)_2 \cdot 8 H_2O$ is dissolved in 455 ml. of  $H_2O$  contained in a silver flask. The solution is briefly refluxed and allowed to stand for 24 hours; the  $SrCO_3$  precipitate is then filtered off. The solution is then brought to a boil, and 35 ml. of a Ni( $ClO_4$ )<sub>2</sub> :  $Sr(ClO_4)_2$  solution (molar ratio ~ 1:4) is added. The latter solution is prepared by adding 25 ml. of  $H_2O$  to 6.5 g. of  $NiCl_2 \cdot 6 H_2O$  and 16 g. of  $SrCO_3$ , and then gradually adding 25 ml. of 70%  $HClO_4$ . To remove HCl, this solution is concentrated until dense  $HClO_4$  fumes are evolved, and then diluted with  $H_2O$  to 35 ml. After addition of the perchlorate solution, the reactor mixture is refluxed in the absence of  $CO_2$ . The  $Sr_2[Ni(OH)_6]$  precipitate is filtered off with suction while the mother liquor is still hot (use small-pore-size glass frit) with thorough exclusion of  $CO_2$ ; it is washed with 35% NaOH at room temperature, and then with absolute methanol. The precipitate is shaken for eight hours with absolute methanol, filtered, and washed with methanol and ether. The product is dried and freed of ether by keeping it for several hours in vacuum in a desiccator containing silica gel.

PROPERTIES:

Gray-green, very fine crystalline powder of unidentifiable crystalline habit; not attacked by half-saturated aqueous  $Sr(OH)_2$  solution (0.35 g. SrO/100 ml. H<sub>2</sub>O); gradually decomposed by H<sub>2</sub>O.

REFERENCE:

R. Scholder and E. Giesler. Unpublished data.

# Sodium Trihydroxostannate (II) Na[Sn(OH)3]

 $SnCl_2 + 2 NaOH = Sn(OH)_2 + 2 NaCl$ 189.6 80.0 152.7 116.9

 $Sn(OH)_2 + NaOH = Na[Sn(OH)_3]$ 152.7 40.0 192.7

Tin (II) hydroxide is prepared by treating a milky solution of 25 g. of  $SnCl_2 \cdot 2 H_2O$  in 1.5 liters of  $H_2O$  with a small excess of approximately 10% ammonia (room temperature), diluting to two liters, allowing the solid to settle, removing the slightly turbid supernatant by aspiration, adding two liters of  $H_2O$ , again removing the supernatant, and then repeating this process 2-3 times. The precipitated  $Sn(OH)_2$  is filtered off on a large Pyrex glass frit of medium pore size, at first without suction, then by slowly applying vacuum; it is then washed until essentially chloride-free. The paste is thoroughly dried by suction, calcined to  $SnO_2$ , and analyzed for  $Sn(OH)_2$  content. Yield of  $Sn(OH)_2 : 85\%$ ;  $Sn(OH)_2$  content of paste: about 50%.

Reagent grade NaOH (35 g.) is dissolved in 23 ml. of  $H_2O$  contained in a wide-mouth 150-ml. Erlenmeyer glass flask. The

solution is cooled to 50-60 °C and the entire  $Sn(OH)_2$  paste is added to it, even though some crystalline NaOH may be present. The container is immediately closed off with a rubber stopper carrying a Peligot tube filled with an alkaline pyrogallol solution. The  $Sn(OH)_2$  after brief shaking dissolves while a small amount of dark SnO separates from the sodium hydroxide solution (the concentration of which is now 50%). The mixture is filtered warm through small-pore-size glass frit. The clear filtrate is protected from air and kept at 0°C; after a few hours, crystallization of the salt is complete. The mixture is carefully warmed to 8°C and filtered through large-pore-size glass frit. To remove the mother liquor still on the crystals, the latter are spread on a clay dish precooled to 0°C, and the dish is kept for 12 hours at 0-3°C in an evacuated desiccator. Yield: 6 g.

PROPERTIES:

Colorless, partly clustered small rods, pointed at the ends. When stored for some time in a closed container (even at 0°C) turns dark because of decomposition; very sensitive to moisture and  $O_2$ . After removal from the clay dish, the product is still contaminated with 0.1-0.2 moles of NaOH per mole. During separation and drying, a small percentage of the Sn (II) is converted to Sn (IV).

REFERENCES:

R. Scholder and R. Pätsch. Z. anorg. allg. Chem. <u>216</u>, 176 (1933);
 R. Scholder and K. Krauss. Unpublished data.

# Sodium Hexahydroxochromate (III)

# Na<sub>3</sub>[Cr(OH)<sub>6</sub>]

 $\begin{array}{rrr} {\rm Cr}({\rm ClO}_4)_3 \ + \ 6 \ {\rm NaOH} \ = \ {\rm Na}_3 [{\rm Cr}({\rm OH})_6] \ + \ 3 \ {\rm NaClO}_4 \\ {\rm 350.4} \ \ 240.0 \ \ 223.1 \ \ 367.4 \end{array}$ 

Since commercial chromic hydroxide always contains an appreciable percentage of carbonate, sodium hexahydroxochromate (III) is best prepared from an aqueous  $Cr(ClO_4)_3$  solution obtained from  $Cr_2O_3 \cdot aq_3$ ; the NaClO<sub>4</sub> formed by reaction with NaOH is sufficiently soluble even in highly concentrated NaOH.

A sample of commercial  $Cr_2O_3 \cdot aq$ . of known  $Cr_2O_3$  content corresponding to 3 g. of  $Cr_2O_3$  is dissolved in the stoichiometric quantity of 20-25% HClO<sub>4</sub>. The solution is concentrated to 25 ml. and filtered; the filtrate is then added to 300 ml. of carbonate-free 51% NaOH. The mixture is refluxed for about 0.5 hours, cooled to

about 120°C, and filtered into a suction flask preheated to 95°C. The dark-green filtrate is transferred to a silver flask protected by a Peligot tube (filled with 50% KOH) and allowed to stand for about four hours in an electric drying oven at 85°C. The precipitated hexahydroxochromate is washed twice with some 40% NaOH (18°C), shaken for 0.5 hours with 80 ml. of 5% methanolic NaOH (18°C), washed several times with the same alkaline methanol solution, and, finally, thoroughly washed with acetone; the acetone is then removed by prolonged vacuum evaporation in a desiccator containing silica gel. The product is very pure (Cr: Na = 1:2.99-3.02). Yield: 5-6 g.

On cooling, the  $Na_3[Cr(OH)_6]$  mother liquor deposits tightly clustered platelets of mixed crystals of hepta- and octahydroxo-chromate (III).

PROPERTIES:

Microcrystalline green powder (well-formed polyhedra). At first soluble in cold  $H_2O$ , affording a clear solution, which after a long time gradually yields a flocculent precipitate of  $\operatorname{Cr}_2O_3 \cdot \operatorname{aq}$ .

**REFERENCE:** 

R. Scholder and R. Pätsch. Z. anorg. allg. Chem. 220, 411 (1934).

# Sodium Hydroxoferrates (III)

These products are obtained by the oxidation of a solution of  $Na_2[Fe(OH)_4]$  in 50% NaOH with O<sub>2</sub>. Under otherwise identical conditions, sodium octahydroxoferrate (III) is formed at 20-25°C, heptahydroxoferrate (III) at 50-60°C, and olive-green oxoferrate (III) at 100-130°C. Boiling 55-60% NaOH yields the red oxoferrate (III), NaFeO<sub>2</sub>.

SODIUM HEPTAHYDROXOFERRATE (III), Na<sub>4</sub>[Fe(OH)<sub>7</sub>] · 2 H<sub>2</sub>O

I. 
$$2 \operatorname{Na}_{2}[\operatorname{Fe}(\operatorname{OH})_{4}] + 4 \operatorname{NaOH} + \frac{1}{2}O_{2} + H_{2}O = 2 \operatorname{Na}_{4}[\operatorname{Fe}(\operatorname{OH})_{7}]$$
  
339.8 160.0 18.0 605.9

A solution of  $Na_2[Fe(OH)_4]$  in 50% NaOH, prepared as described on p. 1686 and cooled to 120 °C, is filtered into an Erlenmeyer flask containing 100 ml. of 50% NaOH. This flask is connected to two wash bottles each containing 50% KOH, and a fast O<sub>2</sub> stream is passed through the solution (kept at 60 °C) for about 12 hours. This causes a gradual discoloration of the greenish-blue solution and simultaneous crystallization. The crystals are filtered off, rapidly washed with 50% NaOH, and dried as a thin layer on a clay dish in an empty desiccator containing silica gel.

The oxidation with  $Br_2$  is more elegant. The flask containing  $Na_2[Fe(OH)_4]$  solution is closed off with a rubber stopper carrying a Peligot tube and a dropping funnel. Then a solution of 2-3 ml. of  $Br_2$  in 10 ml. of  $CCl_4$  is added dropwise at 50-60°C with vigorous agitation until the iron solution just turns colorless. An excess of  $Br_2$  must be avoided. The mixture is allowed to stand for two hours at the same temperature and filtered.

II. Freshly precipitated, thoroughly washed  $Fe_2O_3 \cdot aq$ . is added to carbonate-free 50% NaOH. An amount of NaOH equal in weight to the water contained in the  $Fe_2O_3$  paste is then added and the latter is dissolved with moderate heating (not to exceed 60°C). The mixture, in a silver Erlenmeyer flask protected with a Peligot tube (filled with 50% KOH), is allowed to stand for several days in an electric drying oven at 70°C. In this manner, the  $Fe_2O_3 \cdot aq$ . is completely converted into the nearly colorless, microcrystalline Na<sub>4</sub>[Fe(OH)<sub>7</sub>]  $\cdot$  2 H<sub>2</sub>O, which is sparingly soluble in concentrated NaOH.

PROPERTIES:

Nearly colorless crystalline powder (beveled, partly clustered platelets); very sensitive to moisture. Instantly decomposed by  $H_2O$  and  $CH_3OH$ , affording  $Fe_2O_3 \cdot aq$ . Unstable even in 30% NaOH (18°C).

# SODIUM OCTAHYDROXOFERRATE (III), Na<sub>5</sub>[Fe(OH)<sub>8</sub>] · 5 H<sub>2</sub>O

Oxidation of  $Na_2[Fe(OH)_4]$  with  $O_2$  in a strongly alkaline solution (see method I above) at 20°C yields fine needles of octahydroxoferrate (III), which is also nearly colorless. When allowed to stand at room temperature for a few days, freshly prepared  $Fe_2O_3 \cdot aq$ . (see method II above) is converted to a large extent, but never completely, to the same salt.

**REFERENCES:** 

R. Scholder. Angew. Chem. <u>49</u>, 255 (1936); R. Scholder and K. Krauss. Unpublished data.

# Barium Hydroxoferrates (III)

Barium hydroxoferrates (III) are prepared by dropwise addition of an  $Fe(ClO_4)_3$ -Ba(ClO<sub>4</sub>)<sub>2</sub> solution to hot NaOH. If the initial

NaOH concentration is 25-39%, a precipitate of the hexahydroxo salt, Ba<sub>3</sub>[Fe(OH)<sub>6</sub>]<sub>2</sub>, is obtained; however, if this NaOH concentration exceeds 42%, the heptahydroxo salt, Ba<sub>2</sub>[Fe(OH)<sub>7</sub>]  $\cdot$  <sup>1</sup>/<sub>2</sub> H<sub>2</sub>O, precipitates.

The starting  $Fe(ClO_4)_3$ -Ba $(ClO_4)_2$  solution (1:3 molar ratio) is obtained by dissolving 3.5 g. of  $Fe_2O_3$  (analytical grade) in a mixture of 35 ml. of 70% HClO<sub>4</sub> and 25 ml. of conc. HCl. To eliminate the HCl, the solution is concentrated until dense fumes of HClO<sub>4</sub> are given off. The resulting solution is added, with agitation, to a slurry of 26 g. of BaCO<sub>3</sub> in 125 ml. of H<sub>2</sub>O; the mixture is then filtered.

#### BARIUM HEXAHYDROXOFERRATE (III), Ba<sub>3</sub>[Fe(OH)<sub>6</sub>]<sub>2</sub>

The starting 33% NaOH is prepared by diluting 180 ml. of carbonate-free 50% NaOH with 140 ml. of  $CO_2$ -free H<sub>2</sub>O in a refined silver flask. The mixture is heated to reflux with exclusion of  $CO_2$ , and 75 ml. of the above  $Fe(ClO_4)_3$ -Ba $(ClO_4)_2$  solution is added dropwise. A white precipitate of Ba<sub>3</sub>[Fe(OH)<sub>6</sub>]<sub>2</sub> forms immediately. The mixture is allowed to reflux for one hour, after which it is cooled to room temperature, suction-filtered through a medium-pore-size glass frit, and washed with a small amount of 33% NaOH. The precipitate is vigorously shaken for a few minutes with 200 ml. of absolute methanol, filtered through a small-pore-size glass frit, and washed with absolute methanol and anhydrous ether. It is then dried for one hour in a vacuum desiccator over silica gel.

PROPERTIES:

White to slightly yellowish hexagonal platelets. Decomposed by  $H_2O$ , affording  $Fe_2O_3 \cdot aq_*$ ; stable in absolute methanol.

#### BARIUM HEPTAHYDROXOFERRATE (III), Ba<sub>2</sub> [Fe(OH)<sub>7</sub>] · <sup>1</sup>/<sub>2</sub> H<sub>2</sub>O

The preparation of this compound is analogous to that of  $Ba_3[Fe(OH)_6]_2$ . However, instead of 33% sodium hydroxide, 400 ml. of 50% NaOH is used.

#### PROPERTIES:

White to slightly yellowish hexagonal platelets. Decomposed by  $H_2O$ , affording  $Fe_2O_3 \cdot aq$ . Prolonged contact with absolute methanol yields a brown solution.

Strontium hexahydroxoferrate (III),  $Sr_3[Fe(OH)_{\theta}]_2$ , can be prepared in a similar way, using 5% NaOH, while strontium hepta-hydroxoferrate (III),  $Sr_2[Fe(OH)_7] \cdot 3 H_2O$  requires 20% NaOH.

REFERENCE:

R. Scholder. W. Zeiss and M. Kreutz. Unpublished data.

# Alkali Aluminates

Depending on the temperature, the following three sodium aluminates crystallize from a solution containing NaOH and  $Al_2O_3$  in equal concentrations:

tetrasodium heptahydroxoaluminate  $Na_4[Al(OH)_7] \cdot 3 H_2O$ monosodium oxohydroxoaluminate I  $Na_2O \cdot Al_2O_3 \cdot 2.67 H_2O = Na_8[Al_8O_4(OH)_{18}]$ 

monosodium oxohydroxoaluminate II  $Na_2O \cdot Al_2O_3 \cdot 2.5 H_2O = Na_4[Al_4O_3(OH)_{10}]$ 

Only monopotassium oxohydroxoaluminate,  $K_2O \cdot Al_2O_3 \cdot 3 H_2O = K_2[Al_2O(OH)_6]$ , is obtained from a potassium aluminate solution.

TETRASODIUM HEPTAHY DRO XOALUMINATE, Na<sub>4</sub> [Al(OH)<sub>7</sub>] · 3 H<sub>2</sub>O

 $\begin{array}{rrrr} Al(OH)_3 &+ 4 \text{ NaOH} &= \text{ Na}_4[Al(OH)_7] \\ \hline & & & (\cdot 3 \text{ H}_2 \text{O}) \\ \hline & & 78.0 & 160.0 & 292.1 \end{array}$ 

Aluminum hydroxide (45 g.) is dissolved in a solution of 130 g. of NaOH in 100 ml. of  $H_2O$  by refluxing one half hour. The solution is slowly cooled to room temperature, allowed to stand for six hours, and only then filtered through a small-pore-size glass frit to remove the considerable amount of Na<sub>2</sub>CO<sub>3</sub> precipitate [commercial Al(OH)<sub>3</sub> often contains a large percentage of carbonate]. The crystallization of sodium aluminate, which usually takes a long time to develop, does not start during this period. The clear filtrate is transferred to a round-bottom glass flask closed off with rubber stoppers carrying an air-tight agitator and a Peligot tube. The flask is immersed in 18°C water and its contents are vigorously stirred for 10-14 hours. A thick crystal slurry is formed; this is dried by suction, spread out in a thin layer on a clay dish, and finally dried in an empty desiccator. Yield: 38 g.

#### PROPERTIES:

Microcrystalline powder; strongly birefringent oblong prisms with beveled end faces. Soluble in  $H_2O$ . Contains (as impurity) 0.2-0.3 moles of NaOH per mole of aluminate.

# MONOSODIUM OXOHYDROXOALUMINATE I, Na<sub>6</sub> [Al<sub>6</sub>O<sub>4</sub>(OH)<sub>16</sub>]

A clear aluminate solution is prepared in the manner described above and stirred for 8-10 hours at 40-45 °C. The crystal slurry is washed with 50% NaOH, covered with methanol, and shaken for 0.5 hours with 150 ml. of methanol. The mixture is filtered, thoroughly washed with a large quantity of methanol followed by acetone, and vacuum-dried over silica gel. Yield: 24 g. Analysis shows Al: Na = 1:1.02-1.04.

PROPERTIES:

Formula weight 636.15. Microcrystalline powder (square plates with beveled edges). Transient solubility in  $H_2O$ .

# MONOSODIUM OXOHYDROXOALUMINATE II, Na<sub>4</sub> [Al<sub>4</sub>O<sub>3</sub>(OH)<sub>10</sub>]

The aluminate solution (see above) is stirred for about six hours at 100-105 °C. Otherwise, the preparation, isolation, purification and drying are the same as described above. Very pure product is obtained.

PROPERTIES:

Formula weight 417.98. Microcrystalline powder (thin polyg-onal platelets).

# MONOPOTASSIUM OXOHYDROXOALUMINATE, K2[A12O(OH)6]

A solution prepared at the boil from 120 g. of KOH, 30 g. of Al(OH)<sub>3</sub>, and 100 ml. of  $H_2O$  is allowed to stand for several hours at room temperature, filtered, seeded with the salt (see below), and shaken for 24 hours. The microcrystalline solid deposit is washed with a small amount of 50% KOH, then with 150 ml. of methanol containing 5% KOH, and finally with acetone; it is then vacuum-dried over silica gel. Yield: 5 g.

Without seeding, the crystallization is delayed for several days. The seeding crystals are obtained by preparing a solution containing 20 g. of KOH, 5 g. of Al(OH)<sub>3</sub>, and 10 ml. of H<sub>2</sub>O, filtering at room temperature, and shaking for 12 hours. This produces an abundant crop (about 6.5 g.) of monopotassium aluminate crystals. These crystals, however, are very small and are difficult to free from the adhering KOH, particularly if the latter is very concentrated.

# PROPERTIES:

Formula weight 250.20. Microcrystalline powder (polyhedra); incompletely soluble in water. Can be obtained very pure.

**REFERENCES**:

R. Fricke and P. Jucaitis. Z. anorg. allg. Chem. <u>191</u>, 129 (1930); R. Scholder, W. Kleeberg and M. Schröder. Naturforschung und Medizin in Deutschland, 1939-1946 (FIAT Review), Vol. 25, Inorg. Chem., part III, p. 141.

# Sodium Hexahydroxostannate (IV)

#### Na<sub>2</sub>[Sn(OH)<sub>6</sub>]

 $Sn(OH)_4 + 2 NaOH = Na_2[Sn(OH)_6]$ 186.7 80.0 266.8

A solution of  $SnCl_4$  in very dilute hydrochloric acid is neutralized to methyl orange with carbonate-free NaOH. The  $SnO_2$  aq. precipitate is filtered off, washed until chloride-free with  $H_2O$ , and added in portions to an excess of concentrated,  $100 \,^{\circ}C$  NaOH, in which it dissolves rapidly, affording a clear solution. The crystalline hexahydroxostannate precipitates after a short time. The crystal slurry is filtered in the absence of  $CO_2$  and washed with 30% NaOH and then several times with ethanol and ether.

PROPERTIES:

Colorless crystalline powder (thin hexagonal leaflets). Readily soluble in  $H_2O$ ; the solubility decreases markedly with temperature (see Reiff and Toussaint). Always contains small amounts of adsorbed NaOH. Very sensitive to  $CO_2$ .

REFERENCES:

H. Zocher. Z. anorg. allg. Chem. <u>112</u>, 1 (1920); R. Reiff and S. M. Toussaint. Ibid. <u>241</u>, 372 (1939).

#### Sodium Hexahydroxoplumbate (IV)

Na<sub>2</sub>[Pb(OH)<sub>6</sub>]

I. ELECTROCHEMICAL METHOD

 $Na[Pb(OH)_3] + NaOH + 2 OH^2 - 2 e = Na_2[Pb(OH)_6]$ 281,2 40.0 355.3

Yellow PbO (analytical grade, 18.5 g.) is dissolved in 300 ml. of boiling 13N NaOH; the solution is suction-filtered through a

small-pore-size glass frit and allowed to cool in a  $CO_{g}$ -free atmosphere. This hydroxoplumbate (II) solution is unstable and on prolonged standing gradually deposits crystalline lead oxide. Hence it should be electrolyzed as soon as it has cooled to room temperature. Sometimes it may be necessary to separate the solution from the precipitated PbO by decantation right after cooling.

A rectangular 300-ml, glass jar covered with a rubber plate forming an air-tight seal is used as the electrolysis cell. Through appropriate openings in the lid the tank is provided with a gas outlet tube, a thermometer, an air-tight stirrer, an anode lead-in wire cemented into a glass tube, and a porous clay cell serving as the cathode space. The entire system must be gas-tight. Smooth platinum electrodes  $(5 \times 5 \text{ cm.})$  are used. The electrolysis is carried out at ambient temperature with a current density of 0.12-0.18 amp./in.<sup>2</sup>, while vigorously stirring the strongly alkaline hydroxoplumbate (II) solution in the anode space. The cathode space contains concentrated NaOH. The plumbate (IV) separates in the form of a white crystalline precipitate. The precipitate is allowed to settle, the clear solution is siphoned off, and the crystal slurry is covered with absolute ethanol. The crystals and liquid are then transferred to a smaller container and repeatedly digested with absolute ethanol until the latter no longer shows an alkaline reaction. The pure white crystals become slightly vellowish on vacuum drying in a desiccator.

# II. CHEMICAL METHOD

 $\begin{array}{rrrr} Pb(CH_{3}COO)_{4} &+& 6 \ NaOH &=& Na_{2}[Pb(OH)_{6}] &+& 4 \ CH_{3}COONa \\ 443.4 & & 240.1 & 355.3 & 328.2 \end{array}$ 

A one-liter, round-bottom glass flask equipped with an airtight stirrer, a dropping funnel and a Peligot tube (filled with 30% KOH), all inserted through rubber stoppers, is charged with 200 ml. of carbonate-free 30% NaOH. The tip of the dropping funnel is inserted into a short glass tube (15 mm. I.D.) to protect it from the splashing NaOH solution. A solution (usually yellowish) of 50 g. of Pb(CH<sub>3</sub>COO)<sub>4</sub> in 200 ml. of K<sub>2</sub>CO<sub>3</sub>-dried chloroform containing 1 ml. of glacial acetic acid (filtered, if necessary) is added drop-by-drop with vigorous stirring. The brown PbO<sub>2</sub>, formed at the site of contact between the drops of the chloroform solution and the NaOH in the flask, dissolves rapidly; after a while, Na<sub>2</sub>[Pb(OH)<sub>6</sub>] begins to precipitate. Following the addition of the chloroform solution, the mixture is stirred until the crystalline suspension is pure white. The precipitate is allowed to settle for several hours; it is then filtered off and washed twice with 30% NaOH and at least five times with methanol containing 1% alkali. During this procedure, atmospheric moisture and  $CO_2$  must be absent. The precipitate is then dried on a clay dish placed in an evacuated desiccator over silica gel. A pure white product is obtained provided processing is rapid. Yield: 33 g.

Should the salt be yellowish, it can be purified as follows. The moist product is added in small portions to 500 ml. of 15% NaOH at 75°C (agitation), making sure that each portion is completely dissolved before adding the next. The small residue is filtered off, 150 g. NaOH is added to the filtrate while it is still hot, and the mixture is vigorously shaken to complete dissolution. After cooling, the pure white crystalline powder is filtered off and washed and dried as described above. Yield: 26 g.

#### PROPERTIES:

Colorless crystalline powder (hexagonal polyhedra); stable in 2% NaOH at  $18^{\circ}$ C; very sensitive to moisture. Discolors after absorbing H<sub>2</sub>O. Always contains some excess NaOH.

**REFERENCES:** 

- I. G. Grube. Z. Electrochem. <u>28</u>, 273 (1922); A. Simon. Z. anorg. allg. Chem. <u>177</u>, 109 (1929).
- II. R. Scholder, Unpublished data.

# Barium Oxohydroxostannate (II)

#### Ba[Sn<sub>2</sub>O(OH)<sub>4</sub>]

2 Na[Sn(OH) <sub>3</sub> ]	$+ Ba(OH)_2$	= Ba[Sn <sub>2</sub> O(OH) <sub>4</sub> ]	 2 NaOH	+	$H_2O$
385.5	171.4	<b>458</b> .8	80.0		18,0

The entire  $Sn(OH)_2$  paste obtained from 25 g. of  $SnCl_2 \cdot 2 H_2O$ (preparation as for  $Na[Sn(OH)_3]$ , p. 1687) is added to a 50°C solution of 60 g. of NaOH in 50 ml. of  $H_2O$ . The mixture is cooled to 30°C and a hot solution of 1 g. of  $Ba(OH)_2 \cdot 8 H_2O$  in 2 ml. of  $H_2O$  is added. The mixture is allowed to stand at this temperature for about one hour (air must be absent) and is then filtered to remove the dark SnO and the precipitated carbonate. The clear filtrate is heated to 65°C and treated with a hot solution of 9 g. of  $Ba(OH)_2 \cdot 8 H_2O$  in 20 ml. of  $H_2O$  (95°C). The greenish-yellow barium oxohydroxostannate precipitates within a few minutes. The supernatant is decanted; the solid is filtered off and covered with 50% NaOH. The salt is then washed with 50 ml. of 2%  $Ba(OH)_2 \cdot 8 H_2O$  solution in methanol, followed by pure, 0°C methanol; it is dried on a clay dish over silica gel. Yield: 5 g. Slightly contaminated with  $BaCO_3$ .

PROPERTIES:

Yellowish microcrystalline powder (plates beveled at ends). Decomposed by  $\rm H_2O_{\bullet}$ 

**REFERENCES:** 

R. Scholder and R. Pätsch. Z. anorg. allg. Chem. <u>216</u>, 176 (1933);
 R. Scholder and K. Krauss. Unpublished data.

## SECTION 3

# Iso- and Heteropoly Acids and Their Salts B. GRÜTTNER AND G. JANDER

## Introduction

#### **ISOPOLY COMPOUNDS**

Compounds with higher aggregated anions, in which the anionforming element occurs at least twice, are termed isopoly compounds. Usually it is the alkali or ammonium salts of the isopolyacids that are synthesized. Compounds with isopolyanions are formed, among others, by boron, silicon, phosphorus, arsenic, sulfur, vanadium, molybdenum and tungsten. They may be prepared in a number of ways, e.g., by fusion of an acid anhydride with an alkali hydroxide, dehydration of acid salts, or treatment of a normal salt with its acid anhydride.

Derivatives of the weaker, oxygen-containing metallic acids, such as those of tungstic, molybdic or vanadic acids, exhibit a quite characteristic behavior, and may therefore be considered as the "classical" isopoly compounds. One property characteristic of these metallic acids is the more or less sharply pronounced hydrolysis of their salts in aqueous solution, particularly in the presence of  $H^+$  ions. The hydrolysis products then undergo, over a period of time, a secondary reaction, combining to more highly aggregated ions, that is, the isopolyanions. For example:

 $\begin{array}{r} 6 \left( \mathrm{WO}_4 \cdot \mathrm{aq.} \right)^{2-} + \ 6 \ \mathrm{H}^+ \rightleftarrows \ 6 \left( \mathrm{HWO}_4 \cdot \mathrm{aq.} \right)^{-} \\ 6 \left( \mathrm{HWO}_4 \cdot \mathrm{aq.} \right)^{-} + \ \mathrm{H}^+ \rightleftarrows \left( \mathrm{HW}_6 \mathrm{O}_{21} \cdot \mathrm{aq.} \right)^{5-} + \ 3 \ \mathrm{H}_2 \mathrm{O}. \end{array}$ 

In addition, the following rules apply to the isopoly compounds of vanadium, molybdenum and tungsten.

Specific isopolyanions of definite degrees of condensation and specific chemical properties predominate in the solution; their existence is a function of the H<sup>+</sup> concentration and their crystalline salts may be isolated if certain conditions are observed. When solid, almost all salts of these isopolyacids contain water of crystallization. In keeping with their structure, these compounds are very sensitive to OH<sup>-</sup>, which rapidly degrades most of them to simple molecular compounds. The free acids cannot be isolated since the presence of excess H<sup>+</sup> causes progressive aggregation until insoluble high-molecular-weight hydrated oxides precipitate. "Metatungstic acid" (dodecatungstic acid) is an exception; its overall chemical and crystallographic behavior places it among the heteropoly compounds; thus, it will be discussed in that section.

The preparation of polyphosphates, polysilicates and polyborates is discussed in sections on the respective elements, e.g., on pages 546 ff., 697 ff., 793 ff., and 704 ff. of this handbook.

## HETEROPOLY COMPOUNDS

Heteropoly compounds are composed not only of the weak, oxygen-containing metallic acids (tungstic, molybdic and vanadic), but also of moderately strong to weak acids of nonmetals, e.g., boric, silicic, phosphoric, arsenic, telluric, etc., acids. Stable heteropoly compounds very frequently show nonmetallic to metallic acid ratios of 1:12, 1:6 or 1:9. Since the heteropoly compounds form under conditions similar to those in which isopoly compounds are obtained, that is, only in solutions containing  $H^+$  ions, it is assumed that the building blocks of the heteropolyanions are isopolyanions [1].

In keeping with their constitution, all compounds of this class are quite unstable in the presence of OH ions and are degraded to the simple metallic and nonmetallic acids. The careful degradation of very complex heteropolyanions by agents such as K<sub>2</sub>CO<sub>3</sub>, which act as weak sources of hydroxyl ions, permits the isolation of several intermediates, but nothing further is yet known concerning the mechanism. The heteropolyacids are somewhat more stable to H<sup>+</sup> ions, so that partial isolation of the free acids is possible. Further characteristic properties of numerous heteropoly compounds include their good crystallizability and their relatively high water content per mole of the solid. No heteropoly compounds lacking water in the anion complex are known. Another peculiarity of this class of compounds is that many of the free acids, as well as their salts, crystallize isomorphously [2]. The free acids are rather basic and the formation of neutral salts occurs only as an exception; generally, only acid salts can be isolated. The free acids are specifically capable of forming heavy, oily addition compounds with ether, even when the latter is in the vapor form; these have only a limited miscibility with water and excess ether and easily decompose [3]. This property is commonly used in the preparation of the crystalline acids (see below).

In general the heteropoly compounds of tungstic acid are more complexed and more resistant to hydrolysis than those of molybdic or even vanadic acid. The stability within the same class of compounds varies depending on the nonmetallic acid, so that compounds with phosphoric and silicic acids are more stable than those containing arsenic acid.

Several recently described compounds of molybdic and tungstic acids with phosphoric acid, where the ratio of Mo (or W)/P is below 3:1 (e.g.,  $2 \text{ MoO}_3 \cdot P_2O_5$  or Na<sub>2</sub>O · 2 WO<sub>3</sub> · P<sub>2</sub>O<sub>5</sub>), are not heteropoly compounds; rather, the Mo or W is bound as a cation [4]. We shall omit their discussion.

**REFERENCES:** 

- 1. G. Jander and K. F. Jahr. Kolloid-Beihefte 41, 297 (1935).
- See, e.g., A. Rosenheim and J. Jaenicke. Z. anorg. Chem. <u>77</u>, 239 (1912); <u>101</u>, 235 (1917); H. Copaux. Ann. Chim. Phys. [8] <u>17</u>, 207 (1909); R. Abegg. Handbuch d. anorg. Chem. [Handbook of Inorganic Chemistry], Vol. IV., part 1, 2nd half, p. 993.
- E. Drechsel. Ber. dtsch. chem. Ges. <u>20</u>, 1452 (1887); A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 224, 250 (1917).
- 4. I. Schulz. Ibid. 281, 99 (1955); 284, 31 (1956).

## General Methods

#### 1. FREE HETEROPOLYACIDS BY THE METHOD OF DRECHSEL

A solution of the sodium salt of a heteropoly acid is concentrated as far as possible (even to the sirupy state), placed in a separatory funnel, and covered with about 1/3 its volume of ether. The funnel is shaken vigorously to saturate the solution with the ether. Ice-cold. conc. (37%) iron-free HCl is now added in small portions, with vigorous shaking after each addition. The liquid must not be allowed to heat up during this step; if necessary, the separatory funnel should be externally cooled with water. The liberated acid immediately forms an adduct with the ether and sinks to the bottom as heavy, oily drops which form a third layer. When this layer clarifies, it is drained into a flask. The reaction is complete when addition of hydrochloric acid does not produce further oily droplets at the ether-solution interface. The oil is treated with about an equal volume of H<sub>2</sub>O, and the ether is driven off by drawing a stream of clean, dry air through the mixture. The residual clear aqueous solution of the acid is placed, until incipient crystallization, in a vacuum desiccator over conc. H<sub>2</sub>SO<sub>4</sub> and then over solid KOH to absorb the still present HCl. Only the 12-tungstic-1-boric acid should be crystallized in desiccator over  $P_2O_5$ , in which case this is done to

prevent decomposition of the heteropolyacid by volatilization of the boric acid. Only hydrochloric acid should be used for extraction since the ether adduct is always capable of absorbing this acid, and the latter can then be removed more readily than either sulfuric or nitric acids.

# 2. FREE HETEROPOLYACIDS VIA ION EXCHANGE

The advantage of this method lies in the high purity of the final product. The starting materials-heteropoly salts prepurified by many recrystallizations and extremely soluble in water-are best prepared by the method given below (see 3). In view of the very pronounced acidity of the heteropolyacids and their frequent sensitivity to reducing agents, it is desirable to use cation exchange resins carrying sulfonic acid groups (e.g., Permutit RS), which exhibit only a strong acid function and have almost no reducing power. The operating conditions depend on the sensi-tivity, quality and quantity of the heteropolyacid to be prepared, and can easily be optimized in preliminary experiments. The following rules of thumb should be observed: The exchange capacity of the resin normally amounts to about 2 meq./cm<sup>3</sup> (bulk volume); it is desirable to work with starting solutions which are as concentrated as possible; the throughput of the solution through the column should be low (approx.  $2-5 \text{ cm}^3/\text{min.}$ ). The free heteropolyacid solutions should be concentrated on a steam bath or in a desiccator and, if needed, crystallized.

The method fails with heteropolysalts whose aqueous solutions exhibit a strong acid reaction. Additional complications arise if salt impurities (NaCl, NaNO<sub>3</sub>, etc.) are present in the solution, since these salts produce HCl,  $HNO_3$ , etc., during passage through the column. A too strongly acid medium hinders the formation of free, crystalline heteropolyacids during concentration of the eluate.

The one advantage of this method has already been mentioned. The disadvantages are that one must begin with pure, crystalline alkali salts (which in some cases can only be obtained by the roundabout route of first preparing the free acid by Drechsel's method), and, in addition, the heteropolyacid solutions obtained by ion exchange are often relatively dilute so that their concentration is time-consuming.

# 3. HETEROPOLYSALTS

If the salts cannot be synthesized from their components or cannot be isolated in pure form, they may be conveniently obtained from moderately concentrated solutions of their acids by supersaturation with metal chlorides, or in better yield and purer form by addition of stoichiometric quantities of the metal carbonate. The carbonate should be added carefully, since an excess will induce decomposition of the heteropolyanion.

REFERENCES:

- 1. E. Drechsel. Ber. dtsch. chem. Ges. <u>20</u>, 1452 (1887); A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 224 (1917).
- Based on unpublished experiments of G. Jander and D. Ertel; L. C. W. Baker, B. Loev and Th. P. McCutcheon. J. Amer. Chem. Soc. <u>72</u>, 2374 (1950); R. Klement. Z. anorg. Chem. <u>260</u>, 267 (1949); F. Hein and H. Lilie. Z. anorg. allg. Chem. <u>270</u>, 45 (1952).
- A. Rosenheim and J. Jaenicke. Ibid. <u>101</u>, 224 (1917); H. Copaux. Ann. Chim. Phys. [8] <u>17</u>, 217 (1909).

# ISOPOLY COMPOUNDS

# Isopolyvanadates

1. The Sodium Salt  $2Na_2O \cdot V_2O_5 \cdot aq$ .

 $2 \operatorname{Na_3VO_4}_{367.9} (\text{aq.}) + 2 \operatorname{HClO_4}_{200.9} = 2 \operatorname{Na_2O} \cdot \operatorname{V_2O_5}_{305.9} (\text{aq.}) + 2 \operatorname{NaClO_4}_{244.9}$ 

A 1.1M Na<sub>3</sub>VO<sub>4</sub> solution is prepared by dissolving  $V_2O_5$  in the stoichiometric quantity of carbonate-free NaOH solution, so that 3 moles of Na are present per mole of V; this corresponds to 100.0 g. of  $V_2O_5$  and 132.0 g. of NaOH per liter. Then, 100 ml. of this solution is acidified by dropwise addition of 24.9 ml. of 4.44N HClO<sub>4</sub> (vigorous mechanical stirring). The solution is briefly heated on a steam bath to achieve equilibrium, whereupon the orange liquid becomes colorless. This is then concentrated in vacuum at 25-30°C. The resulting crystals are filtered and washed with some water.

SYNONYM:

Sodium pyrovanadate.

PROPERTIES:

Colorless, water-soluble crystals. Water content: 15 moles of  $H_2O/mole$ . In keeping with its molecular weight, should be considered a salt of a divanadic acid  $H_4(V_2O_7 \cdot aq_*)$ .

<sup>\*</sup>The formula weights given here and subsequently refer to the anhydrous compound.

**REFERENCES:** 

- G. Jander and K. F. Jahr. Z. anorg. allg. Chem. <u>211</u>, 53 (1933); Kolloid-Beihefte <u>41</u>, 35 (1935).
- 2. The Sodium Salt  $Na_2O \cdot V_2O_5 \cdot aq$ .

 $\begin{array}{rrrr} 2 \text{ Na}_{3}\text{VO}_{4} \text{ (aq.)} + 4 \text{ HClO}_{4} = \text{ Na}_{2}\text{O} \cdot \text{V}_{2}\text{O}_{5} \text{ (aq.)} + 4 \text{ NaClO}_{4} \\ 367.9 & 401.9 & 243.9 & 489.8 \end{array}$ 

A 0.812M solution of  $Na_3VO_4$  is prepared as in (1) but using 73.9 g. of  $V_2O_5$  and 97.5 g. of NaOH per liter. Then, 100 ml. of this solution is treated with 57.5 ml. of 2.54N HClO<sub>4</sub> (dropwise addition with stirring) and briefly heated on a steam bath until colorless. The mixture is then concentrated in vacuum at 25-30°C. After filtration the crystals are washed with some water.

SYNONYM:

Sodium metavanadate.

PROPERTIES:

Colorless, water-soluble crystals. Water content: 3 moles of  $H_2O/mole$ . In keeping with its molecular weight, should be considered a salt of a tetravanadic acid  $H_6(V_4O_{13} \cdot aq)$ . Many authors also consider it as the derivative of a trivanadic acid  $H_3(V_3O_9 \cdot aq)$ .

**REFERENCES:** 

- G. Jander and K. F. Jahr. Z. anorg. allg. Chem. <u>211</u>, 53 (1933); Kolloid-Beihefte <u>41</u>, 35 (1935).
- 3. The Sodium Salt  $3 \operatorname{Na_2O} \cdot 5 \operatorname{V_2O_5} \cdot \operatorname{aq}$ .

 $\begin{array}{c} 10 \text{ Na}_{3}\text{VO}_{4} \text{ (aq.)} + 24 \text{ HClO}_{4} = 3 \text{ Na}_{2}\text{O} \cdot 5 \text{ V}_{2}\text{O}_{5} \text{ (aq.)} + 24 \text{ NaClO}_{4} \\ 1839.4 \quad 2411.3 \quad 1095.5 \quad 2938.8 \end{array}$ 

The 0.812M Na<sub>3</sub>VO<sub>4</sub> solution (150 ml.) is prepared as in (2) and 381 ml. of 0.8M HClO<sub>4</sub> is added dropwise with stirring. The mixture is then allowed to stand for about 14 days in a closed flask. At first it becomes dark red, changing to a permanent brighter orange-red in the course of time. It is concentrated in vacuum at 25-30°C. The crystals which form are washed with some water.

PROPERTIES:

Small, hexagonal orange-red platelets with beveled edges, or thin rhombohedra. Grinding changes the crystals into a bright yellow powder. Soluble in  $H_2O$ . Water content: 22 moles of  $H_2O$ /mole. In keeping with the molecular weight, should be considered as a salt of a pentavanadic acid  $H_7(V_5O_{16} \cdot aq.)$  [G. Jander and K. F. Jahr, Kolloid-Beihefte <u>41</u>, 35 (1935)]. Regarded by many authors as a mixture of salts of different basicity, all of them derivatives of hexavanadic acid  $H_4(V_6O_{17} \cdot aq.)$  [see P. Souchay and G. Carpeni, Bull. Soc. Chim. France (5) <u>13</u>, 160 (1946); A. Rosenheim, Z. anorg. allg. Chem. <u>96</u>, 139 (1916)].

REFERENCE:

G. Jander and K. F. Jahr. Z. anorg. allg. Chem. 211, 53 (1933).

4. The Potassium Salt  $K_2O \cdot 3 V_2O_5$ 

This salt is obtained from solutions of commercial potassium "metavanadate"  $(1.04 \text{ K}_2 \text{ O} \cdot \text{V}_2 \text{ O}_5 \cdot 0.78 \text{ H}_2 \text{ O})$  by addition of 1.4 moles of acetic acid per mole of vanadate.

Commercial potassium "metavanadate" (7 g.) is dissolved in 25 ml. of water in a beaker placed in a large heating bath (75°C). The hot solution, which is about 2M in vanadium, is then treated with 70 ml. of 1M acetic acid added from a burette whose tip is immersed in the vanadate solution (vigorous mechanical stirring). The acetic acid is introduced at the rate of 1 ml./minute. The red solution is allowed to remain in the heating bath until the latter cools to room temperature (about 15 hours). The clear solution is then cooled to  $0^{\circ}$ C to induce crystallization. The crystals are filtered off and washed with some ice-cold water, then with acetone.

PROPERTIES:

Formula weight 545.7. Orange-red rhombic crystals or hexagonal platelets, sometimes rather large double pyramids. In view of its chemical behavior and molecular weight, this salt also should be regarded as the salt of pentavanadic acid  $H_5(V_5O_{15} \cdot aq.)$ . A part of the vanadium is said to be bound cationically, the salt thus having the formula  $K_2(VO)[V_5O_{15}]$ .

REFERENCE:

K. F. Jahr and G. Jander. Z. anorg. allg. Chem. 220, 204 (1934).

# Isopolyniobates

Three types of anions exist in aqueous solutions of alkali niobates; these are in reversible equilibria with each other and their ranges of stability depend on the pH of the solution. All alkali isopolyniobates are salts of the hypothetical hexaniobic acid H<sub>8</sub>Nb<sub>6</sub>O<sub>19</sub>. The general method of preparation of those salts in which six to eight H<sup>+</sup> are replaced by M<sup>+</sup> consists of fusion of Nb<sub>2</sub>O<sub>5</sub> with alkali hydroxide or carbonate, solution of the fused cake in H<sub>2</sub>O, and concentration to obtain crystals. The orthoniobate  $M_3NbO_4$  (M = alkali cation) formed in the melt is irreversibly converted to an isopolyniobate by treatment with water. The solubility of the alkali isopolyniobates in water is strongly dependent on the size of the cation. Thus, Li and Na salts invariably dissolve with difficulty (especially in the presence of excess Li<sup>+</sup> or Na<sup>+</sup>), while K, Rb and Cs salts are readily to very readily soluble. The solutions are strongly alkaline. All alkali niobate solutions are very sensitive to acids; even small amounts of mineral acids produce irreversible clouding of the solutions or precipitates of Nb<sub>2</sub>O<sub>5</sub> · aq. On heating to over 300°C. all alkali isopolyniobates lose water irreversibly to give the anhydrous, insoluble alkali metaniobates MNbO3.

 $M_8[Nb_6O_{19} \cdot aq.]$  or  $4M_2O \cdot 3Nb_2O_5 \cdot aq.$  (8:6 type)

$$\begin{array}{l} 8 \text{ MOH } + \ 3 \ \mathrm{Nb}_2\mathrm{O}_5 \ = \ 6 \ \mathrm{M}_3\mathrm{NbO}_4 \ + \ 9 \ \mathrm{H}_2\mathrm{O} \ (\mathrm{fusion}) \\ \\ 6 \ \mathrm{M}_3\mathrm{NbO}_4 \ + \ 5 \ \mathrm{H}_2\mathrm{O} \ = \ \mathrm{M}_8 \ [\mathrm{Nb}_6\mathrm{O}_{19} \cdot \mathrm{aq.}] \ + \ 10 \ \mathrm{MOH} \end{array}$$

To prepare the K salt,  $Nb_2O_5$  and KOH (mole ratio ~1:20, weight ratio ~4:17) are heated in a silver or alumina crucible until a clear melt is obtained. The melt is cooled, ground and dissolved in  $H_2O$ . The solution is decanted to remove any insoluble matter which may be present, then concentrated in vacuum over conc.  $H_2SO_4$  until formation of crystals. These are washed with some water and dried on filter paper.

The corresponding Na salt is obtained from aqueous solutions of the K salt by treatment with NaOH (stirring). The fine, crystalline, white precipitate is filtered off, washed with water, alcohol and ether, and dried.

#### PROPERTIES:

Well-crystallized salts; the water content varies somewhat depending on the method of preparation. The large, transparent crystals of the K salt effloresce when stored under sharply desiccating conditions; they then become cloudy, but retain their good solubility in water. Depending on the conditions of precipitation (hot or ice-cold solutions), the 8:6 sodium salt gives differing crystalline forms (needles or leaflets), which also differ in their water content.

In aqueous solutions, salts of the type  $M_{\theta}[Nb_{\theta}O_{19} \cdot aq.]$  are stable only at pH > 13.

 $M_7[HNb_6O_{19} \cdot aq.]$  or  $7M_2O \cdot 6Nb_2O_5 \cdot aq.$  (7:6 type)

 $M_8 [Nb_6O_{19} \cdot aq] + H_2O = M_7 [HNb_6O_{19} \cdot aq] + MOH$ 

Two recrystallizations of the 8:6 sodium salt from  $H_2O$  afford  $Na_7[HNb_6O_{19} \cdot aq.]$ .

Another method of preparation starts with the fusion of Nb<sub>2</sub>O<sub>5</sub> with Na<sub>2</sub>CO<sub>3</sub> (mole ratio ~ 1:4, weight ratio ~ 5:8); the cooled melt is ground, treated with a large amount of H<sub>2</sub>O, and stirred for several hours. Since the solubility of sodium niobate is poor, only the excess Na<sub>2</sub>CO<sub>3</sub> dissolves in this operation. The residue is recrystallized from H<sub>2</sub>O to give pure 7:6 sodium niobate.

The corresponding K salt is best prepared by addition of alcohol to solutions which contain 10 weight percent or more of pure 8:6 potassium niobate.

PROPERTIES:

The 7:6 sodium niobate forms long crystalline needles; water content: 32 moles of  $H_2O/mole$ .

The 7:6 potassium niobate precipitated with alcohol readily loses its water of crystallization and forms lower hydrates; e.g., at 100°C, it gives the penta- or tetrahydrate, and at 150°C, the dihydrate.

In aqueous solutions, salts of the type  $M_7[HNb_6O_{19} \cdot aq.]$  are stable only in the pH range of 9 to 13.

 $(M_{6}[Nb_{6}O_{18} \cdot aq.])_{n}$  or  $(6 M_{2}O \cdot 6 Nb_{2}O_{5} \cdot aq.)_{n}$  (6:6 type)

 $n M_7 [HNb_6O_{19} \cdot aq] + n H_2O = (M_6 [Nb_6O_{18} \cdot aq])_n + n MOH$ 

A 2-4% aqueous solution of 8:6 potassium niobate (or a concentrated solution of 7:6 potassium niobate) is treated by dropwise addition of an equal volume of methyl alcohol (cooling in ice, vigorous mechanical stirring). The product is an amorphous, flocculent hydrated potassium niobate.

It is filtered, washed with 50% methyl alcohol, and dried under mild conditions.

PROPERTIES:

Pure white powder. Readily soluble in water. The water content varies depending on the conditions of preparation. The hydrated metaniobate is stable only in aqueous solutions of pH < 8 (probably as far as the region of the isoelectric point, which occurs at pH ~4.5). At higher pH values, changes first to the 7:6 type (pH 9-13), and then to the 8:6 type (pH > 13). Based on diffusion measurements, the anion ( $[Nb_6O_{18} \cdot aq.]^6$ )<sub>n</sub> has an ionic weight ~3000 (n = 3-4), so that the designation in the heading of this section is preferred to the formulas  $M_2O \cdot Nb_2O_5 \cdot aq$ . or  $MNbO_3 \cdot aq$ . which are sometimes encountered.

REFERENCES:

G. Jander and D. Ertel. J. Inorg. Nuclear Chem. <u>14</u>, 71, 77, 85 (1960); A. V. Lapitskiy and V. I. Spitsyn. Zh. Prikl. Khim. <u>26</u>, 101 (1953); F. Windmaisser. Österr. Chemiker-Ztg. <u>45</u>, 201 (1942); P. Süe. Ann. Chimie [11] <u>7</u>, 493 (1937).

# Isopolytantalates

As far as the general method of preparation of alkali isopolytantalates, their water solubility and their thermal behavior are concerned, the introductory remarks made in the section on isopolyniobates apply here as well. However, the composition of the alkali isopolytantalates, i.e., the base:acid ratio, is not yet completely clear. While some authors find hexatantalates (8:6) exclusively, others have established that only pentatantalates (7:5) exist, and still others insist that both types of compounds occur together, and are possibly related to each other via a region in which only one exists.

For this reason, we have given here several procedures taken from the original references.

 $K_7[Ta_5O_{16} \cdot aq.]$  or  $7K_2O \cdot 5Ta_2O_5 \cdot aq.$  (pentatantalate, 7:5 type)  $K_8[Ta_6O_{19} \cdot aq.]$  or  $4K_2O \cdot 3Ta_2O_5 \cdot aq.$  (hexatantalate, 8:6 type)

Either  $Ta_2O_5$  and KOH (mole ratio ~1:20, weight ratio ~2:5) or  $Ta_2O_5$  and  $K_2CO_3$  (mole ratio ~1:4) are heated in a silver or alumina crucible (or a platinum vessel) until a clear melt is obtained. The melt is cooled, ground and dissolved in  $H_2O$ . The solution is decanted from any insoluble matter and concentrated in vacuum over conc.  $H_2SO_4$  until crystallization occurs. The crystals are rinsed with  $H_2O$  and dried on filter paper. The crystal size increases with the excess alkali hydroxide or carbonate present in the mother liquor.

PROPERTIES:

Hexagonal prismatic columns up to 1 cm. long, with blunt edges; effloresce when stored under sharply desiccating conditions.

Readily soluble in water, giving a strong alkaline reaction. The content of water of crystallization varies.

**REFERENCES:** 

G. Jander and H. Schulz. Z. anorg. allg. Chem. <u>144</u>, 233 (1925);
 G. Jander and D. Ertel. J. Inorg. Nuclear Chem. 3, 139 (1956);
 F. Windmaisser. Z. anorg. allg. Chem. <u>248</u>, 283 (1941).

Na<sub>7</sub> [Ta<sub>5</sub>O<sub>16</sub> · aq.] or 7Na<sub>2</sub>O · 5Ta<sub>2</sub>O<sub>5</sub> · aq. (pentatantalate)

A mixture of  $Ta_2O_5$  and NaOH (mole ratio 1:5, weight ratio 11:5) is melted. The melt is cooled, ground, dissolved in  $H_2O$  and treated in the cold (stirring) with 0.1N NaOH. Pure white Na pentatantalate precipitates. It is washed with  $H_2O$ , alcohol and ether, and dried. Sodium pentatantalate also forms when the aqueous solution of the melt is evaporated at 85°C.

Alternate method: The same salt is obtained by treatment of a hot potassium tantalate solution with hot aqueous NaOH.

PROPERTIES:

Small prismatic needles;  $d^{20}$  3.78. Water content: 22 moles of H<sub>2</sub>O/mole; moderately soluble in water. The pH of a 1% solution is 8.48.

REFERENCES:

V. I. Spitsyn and N. N. Shavrova. Zh. Obshch. Khim. <u>26</u>, 1258 (1956); G. Jander and D. Ertel. J. Inorg. Nuclear Chem. <u>3</u>, 139 (1956).

Na8[Ta<sub>6</sub>O<sub>19</sub> · aq.] or  $4Na_2O \cdot 3Ta_2O_5 \cdot aq$ . (hexatantalate)

The melt obtained by fusion of  $Ta_2O_5$  and NaOH (mole ratio 1:5, weight ratio 11:5) is cooled, ground and treated with ten times its weight of cold  $H_2O$  to remove excess alkali. The residue is dissolved in  $H_2O$  at 80°C and concentrated at 50°C.

PROPERTIES:

Small leaflets;  $d^{20}$  3.58. Water content: 33 moles of  $H_2O/mole$ ; moderately soluble in water; pH of a 1% solution = 8.58. Goniometric measurements indicate that this Na hexatantalate belongs to the hexagonal system.

REFERENCES:

V. I. Spitsyn and N. N. Shavrova. Zh. Obshch. Khim. <u>26</u>, 1258, 1262 (1956).

#### **is**opolyarsenates

Sodium Hydrogen Triarsenate Na<sub>3</sub>H<sub>2</sub>As<sub>3</sub>O<sub>10</sub>

 $\begin{array}{ll} 3 \text{ NaH}_2 \text{AsO}_4 \cdot \text{H}_2 \text{O} \rightarrow \text{ Na}_3 \text{H}_2 \text{As}_3 \text{O}_{10} \ + \ 5 \ \text{H}_2 \text{O} \\ \\ 545.8 \ (\text{incl. } \text{H}_2 \text{O}) & 455.7 & 90.1 \end{array}$ 

This salt is formed on dehydration of  $NaH_2AsO_4 \cdot H_2O$ . Very slow heating of the starting material yields several intermediate products ( $NaH_2AsO_4$  and  $Na_2H_2As_2O_7$ ), which transform above 135°C to the triarsenate  $Na_3H_2As_3O_{10}$ . The last is stable up to 230°C. Rapid heating of the starting  $NaH_2AsO_4 \cdot H_2O$  to temperatures above 96°C yields the triarsenate directly.

The best method of preparation is to place about 10 g. of  $NaH_2AsO_4 \cdot H_2O$  (see p. 602) in a weighing bottle and heat it to constant weight (about 25 hours) in an electric furnace at 135°C.

PROPERTIES:

Absorbs H<sub>2</sub>O from air at room temperature; after several intermediate stages, NaH<sub>2</sub>AsO<sub>4</sub> · H<sub>2</sub>O is finally regenerated. Immediately hydrated to the orthoarsenate upon solution in water. Considered by Thilo and Plaetschke to be the doubly acid salt of the pentabasic triarsenic acid H<sub>5</sub>As<sub>3</sub>O<sub>10</sub> = As<sub>2</sub>O<sub>5</sub> · <sup>5</sup>/<sub>3</sub> H<sub>2</sub>O. For the preparation of As<sub>2</sub>O<sub>5</sub> · <sup>5</sup>/<sub>3</sub> H<sub>2</sub>O, see this handbook, p. 601.

REFERENCE:

E. Thilo and I. Plaetschke. Z. anorg. Chem. 260, 315 (1949).

#### Isopolychromates

Potassium Trichromate  $K_2O \cdot 3CrO_3$ 

This salt is formed on careful evaporation of an aqueous solution of  $K_2Cr_2O_7$  and excess  $CrO_3$ .

A solution of 11.0 g. of  $K_2Cr_2O_7$  and 17.4 g. of  $CrO_3$  (mole ratio  $K_2O:CrO_3 = 1:6.66$ ) in 22.0 ml. of water is prepared at 60°C, a temperature at which the solution is saturated. Evaporation at 60°C yields deep red crystals. The liquid is evaporated to about

13 ml. and then decanted rapidly while still warm. The crystals are dried by pressing on filter paper. Yield: about 7.8 g.

PROPERTIES:

Formula weight 394.2. Deep red prisms, containing no water of crystallization; decomposes on solution in water. Stable in solutions only in the presence of excess  $CrO_3$  or conc.  $HNO_3$ .

Potassium Tetrachromate  $K_2O \cdot 4CrO_3$ 

This salt is obtained from aqueous solutions of  $K_2Cr_2O_7$  in the presence of a large excess of  $CrO_3$ . The evaporation should not be carried too far.

A saturated solution of 15.67 g. of  $K_2Cr_2O_7$  and 43.43 g. of  $CrO_3$  (mole ratio  $K_2:CrO_3 = 1:10.15$ ) in 40.9 ml. of water is prepared at 60°C and concentrated at this temperature to about 10 ml. The nascent crystals are separated and dried as described in the case of the trichromate. Yield: about 13 g.

PROPERTIES:

Formula weight 494.2. Brownish red tablets, containing no water of crystallization; decomposes on solution in water. Stable in solutions only in the presence of excess  $CrO_3$  or conc. HNO<sub>3</sub>.

**REFERENCES:** 

E. Jäger and G. Krüss. Ber. dtsch. chem. Ges. <u>22</u>, 2040 (1889);
F. A. H. Schreinemakers. Z. phys. Chem. <u>55</u>, 71 (1906). Checked by the present authors.

# **Isopolymolybdates**

The compounds described below should be considered derivatives of a hexamolybdic acid  $H_{\mathfrak{G}}(Mo_{\mathfrak{G}}O_{\mathfrak{21}}\cdot aq.)$  [see G. Jander and K. F. Jahr, Kolloid-Beihefte <u>41</u>, 27 (1935)].

The Sodium Salt  $5Na_20 \cdot 12MoO_3 \cdot aq$ .

 $\begin{array}{rl} 12 \text{ MoO}_3 \,+\, 10 \text{ NaOH} \,=\, 5 \text{ Na}_2 \text{O} \cdot 12 \text{ MoO}_3 \,(\text{aq.}) \\ 1727.4 & 400.0 & 2037.3 \end{array}$ 

Sodium hydroxide (8 g.) is dissolved in 100 ml. of hot  $H_2O$ , and 29 g. of  $MoO_3$  is added. The pH of the cooled, clear solution (filtered, if necessary) is about 5. It is evaporated in a vacuum

desiccator over  $H_2SO_4$  to 3/4 to 2/3 of its original volume. The compound precipitates in the form of a slurry, which is filtered and washed with some  $H_2O_2$ .

SYNONYM:

Sodium paramolybdate.

PROPERTIES:

According to Rosenheim, large, lustrous monoclinic prisms which effloresce easily. Water content: 38 moles of  $H_2O/mole$ . Soluble in  $H_2O$ . In our own experiments, evaporation in vacuum or on a steam bath gave a granular white mass, which was not significantly soluble in water either after drying or when freshly prepared and moist.

REFERENCE:

A. Rosenheim. Z. anorg. allg. Chem. <u>96</u>, 143 (1916). Checked by the present authors.

The Ammonium Salt  $5(NH_4)_2 O \cdot 12 M_0 O_3 \cdot aq$ .

The reaction vessel is a porcelain dish. It contains 20 g of  $MoO_3$ , covered with 230 ml. of conc. ammonia. The solution is gently evaporated on a steam bath (solution temperature 60-70°C) until the excess  $NH_3$  is removed and the first crystals form (this occurs upon concentration to about 1/5 of the original volume). The concentrate is cooled and the crystals are filtered off. Yield: about 20 g.

PROPERTIES:

Formula weight 1987.8. According to Rosenheim, large, clear, colorless hexagonal prisms, moderately soluble in  $H_2O$ . Water content: 7 moles of  $H_2O$ /mole. This product is the ammonium molybdate of commerce. Its aqueous solution gives an acid reaction and the compound undergoes hydrolytic cleavage on prolonged boiling. Our own experiments yielded small, white crystals, soluble in hot  $H_2O$ .

REFERENCE:

A. Rosenheim. Z. anorg. allg. Chem. <u>96</u>, 141 (1916). Checked by the present authors.

The Sodium Salt  $Na_2O \cdot 4MoO_3 \cdot aq$ .

$$\begin{array}{rl} 4 \operatorname{Na_2MoO_4}(\mathrm{aq.}) + 6 \operatorname{HCl} = \operatorname{Na_2O} \cdot 4 \operatorname{MoO_3}(\mathrm{aq.}) + 6 \operatorname{NaCl} \\ 823.8 & 218.8 & 637.8 & 350.7 \end{array}$$

A solution of 9.3 g. of  $Na_2MoO_4 \cdot 2 H_2O$  in about 8 ml. of hot  $H_2O$  is treated, while still hot, with 11 ml. of 5.5N HCl added dropwise from a burette. The initial precipitate redissolves, giving a yellowish solution. The liquid, in a stoppered Erlenmeyer flask, is left in a cool place to crystallize. A crystalline crust appears after 24 hours and its thickness increases in the course of the next few days. The crystals are filtered off, washed three times with cold water, and dried by drawing air through the crystal layer. Yield: 6 g.

SYNONYM:

Sodium metamolybdate.

**PROPERTIES**:

Relatively long needles, partially pulverized when touched. Moderately soluble in cold  $H_2O$ , very soluble in hot. Water content: 6 moles of  $H_2O/mole$ .

REFERENCE:

G. Wempe. Z. anorg. Chem. <u>78</u>, 302 (1912). Checked by the present authors.

#### Isopolytungstates

The compounds described below are derivatives of a hexatungstic acid  $H_6(W_6O_{21} \cdot aq.)$  [see G. Jander and K. F. Jahr, Kolloid-Beihefte <u>41</u>, 18 (1935)].

The Sodium Salt  $5 \operatorname{Na_2O} \cdot 12 \operatorname{WO_3} \cdot \operatorname{aq}$ .

$$12 \operatorname{Na_2WO_4}(\operatorname{aq.}) + 14 \operatorname{HCl} = 5 \operatorname{Na_2O} \cdot 12 \operatorname{WO_3}(\operatorname{aq.}) + 14 \operatorname{NaCl}_{3526.9} 510.6 3093.0 818.3$$

A solution of 20 g. of  $Na_2WO_4 \cdot 2 H_2O$  in 40 ml. of hot  $H_2O$  is neutralized to litmus with 2N HCl. About 23.5 ml. of HCl is required (about 1.2 moles of HCl per mole of  $Na_2WO_4$ ). The pH of the solution is then 6.8. The salt is allowed to crystallize in a vacuum desiccator at room temperature over  $H_2SO_4$ .

SYNONYM:

Sodium paratungstate.

PROPERTIES:

Large transparent or milky-white trichlinic crystals. Water content: 28 moles of  $H_2O/mole$ . Readily soluble in water. Other hydrates exist at higher temperatures. The recent views on the complex processes involved in the formation of paratungstates are given by Jander and Krüerke.

**REFERENCES**:

A. Rosenheim. Z. anorg. allg. Chem. <u>96</u>, 160 (1916); C. Scheibler.
 J. prakt. Chem. <u>83</u>, 284 (1861). Checked by the present authors.
 G. Jander and U. Kruerke. Z. anorg. allg. Chem. <u>265</u>, 244 (1951).

The Ammonium Salt  $5(NH_4)_2 \mathbf{0} \cdot 12 WO_3 \cdot aq$ .

Hydrated tungstic acid is dissolved in excess ammonia, and the solution is concentrated on a steam bath or at room temperature, whereby the excess ammonia evaporates.

SYNONYM:

Ammonium paratungstate.

PROPERTIES:

Formula weight 3042.72. Microscopically small, rectangular tablets when the solution is evaporated at high temperatures. Water content: 7 moles of  $H_2O/mole$ . A different hydrate exists at room temperature and below. Rather sparingly soluble in  $H_2O$ . On prolonged boiling in aqueous solution, the salt is hydrolytically decomposed and loses  $NH_3$ .

REFERENCE:

A. Rosenheim. Z. anorg. allg. Chem. <u>96</u>, 158 (1916).

The Zinc Salt  $5 Zn O \cdot 12 WO_3 \cdot aq$ .

 $5 \operatorname{Na_2O} \cdot 12 \operatorname{WO_3} (\operatorname{aq.}) + 5 \operatorname{ZnSO_4} = 5 \operatorname{ZnO} \cdot 12 \operatorname{WO_3} (\operatorname{aq.}) + 5 \operatorname{Na_2SO_4}$ 3093.01 807.2 3189.94 710.3

A solution is prepared by heating a mixture of 7.8 g. of sodium paratungstate (prepared as above) and 70 ml. of water. This mixture is then reacted with a warm, saturated solution of 2.4 g. of

 $ZnSO_4 \cdot 7 H_2O$  in 15 ml. of water. A white precipitate forms, partially redissolves, and then settles out. It is suction-filtered and dried over conc.  $H_2SO_4$ .

SYNONYM:

Zinc paratungstate.

PROPERTIES:

White needles or fine, crystalline precipitate; poor solubility in  $H_2O$ . Water content: 35 moles of  $H_2O$ /mole.

REFERENCE:

A. Rosenheim. Z. anorg. allg. Chem. <u>96</u>, 162 (1916). Checked by the present authors.

# Isopolysulfates

Potassium Trisulfate K2S3O10

I.

$$K_2SO_4 + 2SO_3 = K_2S_3O_{10}$$
  
174.3 160.1 334.4

According to Baumgarten and Thilo,  $K_2S_3O_{10}$  may be prepared from  $K_2SO_4$  by treatment with SO<sub>3</sub>. The apparatus used is shown in Fig. 342.



Fig. 342. Preparation of potassium trisulfate. a' flask with oleum; b reactor tube, here shown surrounded by an electric tubular furnace; d flask with conc. H<sub>2</sub>SO<sub>4</sub>; I-3 stopcocks. The reactor tube c is longer in relation to other parts of the apparatus than shown.

All parts of the glass apparatus are connected by ground joints. The joints are sealed with a paste made from 10 g. of powdered talc (preboiled several times in HCl) and 14 g. of anhydrous phosphoric acid. Stopcock plugs 1 and 2 should be of as large a

diameter as possible; stopcock 3 may be of normal size. After thorough drying of all parts of the apparatus a boat of quartz, glazed porcelain, or platinum containing finely powdered, ignited  $K_2SO_4$  is inserted into reaction tube b; following this, adapter c and flask d (containing 96.5%  $H_2SO_4$ ) are attached. Stopcock 1 is closed, and the section b-d is evacuated with an aspirator. Stopcock 3 is then closed, and the apparatus is left to dry for a while. Flask a is then filled with 70% oleum, a few glass beads are added, stopcock 1 is reopened, and the entire apparatus is rapidly evacuated with the aspirator, so that the oleum evolves. Stopcocks 2 and 3 are then closed. The flask containing the oleum is heated to about 110°C in a sulfuric acid bath until enough SO<sub>3</sub> distills and condenses in tube b to entirely surround the boat with the liquid. Flask *a* is then allowed to cool somewhat and stopcock 1 is closed. Tube b (between 1 and 2) is heated externally to  $50-53^{\circ}$ C, using a sheet iron heating trough lined with asbestos, and covering the top with asbestos. The trough is heated with a row burner, while the stopcocks and flasks are insulated with asbestos to prevent heating. The 50-53°C temperature desired is measured in the space between the reactor tube and the trough. Alternatively, an electric furnace may be used, as shown in the figure.

In the next two hours, the  $K_2SO_4$  will sinter, become a slurry and finally convert to a clear liquid. If the quantity of  $SO_3$  present is insufficient, the melt may resolidify. At the end of the reaction, stopcock 2 is reopened, flask d is cooled in ice, and the excess  $SO_3$  is distilled onto the cooled  $H_2SO_4$ , first at room temperature and finally by heating the reaction tube two hours at 100°C. The apparatus is allowed to cool and stopcock 3 is opened; little or no fuming should then occur.

п.

$$2 \operatorname{KClO_4}_{277.1} + 3 \operatorname{SO_3}_{34.4} = \operatorname{K_2S_3O_{10}}_{10} + \operatorname{Cl_2O_7}_{182.9}$$

According to Lehmann and Krüger, the reaction of SO<sub>3</sub> with KClO<sub>4</sub> produces potassium trisulfate and Cl<sub>2</sub>O<sub>7</sub>; the latter dissolves in the excess SO<sub>3</sub>. The residue obtained on vacuum evaporation of the excess SO<sub>3</sub> and the Cl<sub>2</sub>O<sub>7</sub> is chlorine-free, stoichiometric K<sub>2</sub>S<sub>3</sub>O<sub>10</sub>.

Extremely dry, fine KClO<sub>4</sub> powder, free of reducing impurities, is placed in the apparatus of Baumgarten and Thilo shown in Fig. 342 and treated with SO<sub>2</sub>-free SO<sub>3</sub> at 25-30°C until the contents of the boat become completely liquid. Excess SO<sub>3</sub> and Cl<sub>2</sub>O<sub>7</sub> are then absorbed in conc. H<sub>2</sub>SO<sub>4</sub> at room temperature, while vacuum is applied at 3.

PROPERTIES:

According to recent studies,  $K_2S_3O_{10}$  is thermally stable up to 110°C. Above this temperature the vapor pressure of SO<sub>3</sub> begins

to increase and the compound decomposes into potassium pyrosulfate and SO<sub>3</sub>. The  $K_2S_3O_{10}$  prepared by the method of Baumgarten and Thilo is a cake which can readily be ground to a fine powder. It absorbs water from the air and converts to an adduct of sulfuric acid and pyrosulfate or hydrogen sulfate. It decomposes immediately (with fizzing) in cold water; in the first stage of decomposition, only 1 mole of SO<sub>3</sub> is evolved and a pyrosulfate of relatively poor solubility is formed.

REFERENCES:

P. Baumgarten and E. Thilo. Ber. dtsch. chem. Ges. <u>71</u>, 2596 (1938); H. A. Lehmann and G. Krüger. Z. anorg. allg. Chem. <u>274</u>, 141 (1973); H. A. Lehmann and A. Kluge. Ibid. <u>264</u>, 120 (1951).

# HETERPOLY COMPOUNDS

## 12-Tungstic Acid-1-Borates

In keeping with their constitution and molecular weight, these compounds should be regarded as salts of 12-tungstic-1-boric acid  $H_5[BO_4(W_3O_3)_4 \cdot aq.]$  [R. Signer and H. Gross, Helv. Chim. Acta <u>17</u>, 1076 (1934)].

The Free Acid  $B_2O_3 \cdot 24WO_3 \cdot aq$ .

The first step involves the preparation of a solution of  $5 \operatorname{Na_2O} \cdot \operatorname{B_2O_3} \cdot 24 \operatorname{WO_3} \cdot \operatorname{aq}$ . from  $\operatorname{Na_2WO_4} \cdot \operatorname{aq}$ . and  $\operatorname{H_3BO_3}$ . A large excess of boric acid is used to bind the alkali of the  $\operatorname{Na_2WO_4}$  and to ensure that the solution remains acidic. The acid can be isolated from the solution of the sodium salt by addition of ether and sulfuric acid according to the method of Drechsel (see p. 1700 f.).

A solution of 100 g. of  $Na_2WO_4 \cdot 2 H_2O$  and 150 g. of  $H_3BO_3$  is prepared in 400-500 ml. of boiling  $H_2O$ . The solution is boiled until a sample deposits no tungstic acid when dil. HCl is added. The solution is cooled, suction-filtered to remove the boric acid and sodium polyborate crystals, reacted again with 70 g. of  $H_3BO_3$ , and concentrated over a free flame. The crystalline mass which separates on cooling is again filtered off and washed with some  $33\% H_2SO_4$ . The mother liquor, which contains  $5 Na_2O \cdot B_2O_3 \cdot 24 WO_3 \cdot aq.$ , is extracted with 2-3 volumes of  $33\% H_2SO_4$  and ether according to the method of Drechsel. For further workup see p. 1701.

**PROPERTIES:** 

#### Formula weight 5634.3. Two forms:

a) Perfectly clear octahedral crystals, initially bright but acquiring a greasy luster and a yellowish cast on storage; otherwise, can be stored for a long time. Water content: 65 or 66 moles of  $H_2O/mole$ . M.p. 45-51°C. Soluble in water.

b) Hexagonal needles, less stable, more apt to become yellow and cloudy. Water content: 53 moles of  $H_2O/mole$ . Soluble in  $H_2O$ . The m.p. cannot be determined, since heating causes decomposition. These crystals were formerly thought to be those of an isoborotungstic acid.

REFERENCE:

A. Rosenheim and J. Jaenicke. Z. anorg. Chem. <u>77</u>, 244 (1912); <u>101</u>, 236 (1917).

The Sodium Salt  $5Na_2O \cdot B_2O_3 \cdot 24WO_3 \cdot aq$ .

The crystalline sodium salt is best prepared from the free acid by addition of the stoichiometric quantity of  $Na_2CO_3$ .

 $\begin{array}{rll} B_{2}O_{3} \cdot 24 \ WO_{3} \ (aq.) \ + \ 5 \ Na_{2}CO_{3} \ = \ 5 \ Na_{2}O \cdot B_{2}O_{3} \cdot 24 \ WO_{3} \ (aq.) \ + \ 5 \ CO_{2} \\ 5635.7 \ 530.0 \ 5945.7 \end{array}$ 

A solution containing 34.4 g. of free 12-tungstic-1-boric acid (equivalent to about 41.6 g. of the hydrated acid) is reacted with 3.1 g. of anhydrous  $Na_2CO_3$ , and concentrated first on a steam bath and then in a desiccator over conc.  $H_2SO_4$ .

PROPERTIES:

White, well-formed octahedra. Soluble in  $H_2O$ . Water content: 58 moles of  $H_2O$ /mole.

REFERENCE:

A. Rosenheim and H. Schwer. Z. anorg. Chem. 89, 236 (1914).

#### 12-Tungstic Acid-1-Silicates

In view of their structure and molecular weight, all compounds of this type should be considered as salts of 12-tungstic-1-silicic acid  $H_4[SiO_4(W_3O_9)_4 \cdot aq.]$  [R. Signer and H. Gross, Helv. Chim. Acta 17, 1076 (1934)].
The Free Acid Si  $\mathbf{O}_2 \cdot 12 \mathbf{WO}_3 \cdot \mathbf{aq}$ .

$$\frac{12 \operatorname{Na_2WO_4}(\mathrm{aq.}) + \operatorname{SiO_2} + 20 \operatorname{HCl} = 2 \operatorname{Na_2O} \cdot \operatorname{SiO_2} \cdot 12 \operatorname{WO_3}(\mathrm{aq.}) + 20 \operatorname{NaCl}}{3526.9 \quad 60.06 \quad 729.4 \quad 2967.1 \quad 1169.0}$$

The free acid is isolated from a solution of  $2 \text{ Na}_2\text{O} \cdot \text{SiO}_2 \cdot 12 \text{ WO}_3 \cdot \text{aq}$ . by the method of Drechsel, that is, extraction with ether and conc. HCl.

A solution of 50 g. of  $Na_2WO_4 \cdot 2 H_2O$  in 400 ml. of cold  $H_2O$ is prepared. It is then treated by dropwise addition of about 27 ml. of 6N HCl, until neutral to litmus. The white precipitate formed during the addition redissolves on swirling the flask. An excess of freshly precipitated silicic acid hydrate is now added to the solution. (The silicic acid is prepared as follows. Commercial sodium silicate is dissolved in a minimum of cold  $H_2O$  and made neutral to litmus by dropwise addition of conc. HCl. After 15 minutes a small excess of acid is added. The solution is decanted and the precipitate is washed once or twice with cold water, which is likewise decanted.)

The mixture of tungstate and silicic acid is boiled for about two hours (the liquid being kept acidic by periodic addition of small amounts of HCl) until a filtered sample of the solution no longer precipitates tungstic acid hydrate on addition of dil. HCl. The solution is filtered to remove undissolved  $SiO_2$  and shaken with ether and conc. HCl. For further workup, see p. 1701. If the free acid is to be used only as starting material for preparation of a salt, the oily adduct may be decomposed at about 40°C, the ether removed by long heating, and the excess hydrochloric acid removed by drawing air through the slowly solidifying residue. The yield is 27 g.; the product is not completely pure.

PROPERTIES:

Formula weight 2842.4. The acid crystallizes at room temperature in colorless, lustrous octahedra; m.p. 53°C. Water content: 32 moles of  $H_2O/mole$ . Readily soluble in  $H_2O$ . Several different crystalline hydrates exist, one of which is denoted the "iso acid."

REFERENCE:

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 240 (1917). Checked by the present authors.

The Potassium Salt  $2K_2O \cdot SiO_2 \cdot 12WO_3 \cdot aq$ .

An aqueous solution of the free acid (about one part by weight of acid to three or four parts of  $H_2O$ ), whose content is determined

1718

by evaporation and ignition of an aliquot, is treated by slow addition of the stoichiometric quantity of solid  $K_2CO_3$  (two moles of  $K_2CO_3$ per mole of acid) while applying heat. The clear solution, which must retain an acidic reaction, is evaporated on a steam bath to 1/2 to 1/4 its volume. Cooling precipitates  $2 K_2O \cdot SiO_2 \cdot 12 WO_3 \cdot$ aq., first in hexagonal prisms and then also as rhombic crystals. The product is recrystallized from hot  $H_2O$ .

**PROPERTIES:** 

The hexagonal, colorless prisms effloresce easily. Water content: 18 moles of  $H_2O/mole$ . Readily soluble in hot  $H_2O$ , somewhat less so in cold. The rhombic crystals are said to be the salt of the so-called "iso acid"; they do not effloresce so rapidly. Water content: 9 moles of  $H_2O/mole$ .

**REFERENCE:** 

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 243 (1917). Checked by the present authors.

## 10-Tungstic Acid-1-Silicates

The Potassium Salt  $7 \text{ K}_2 \text{ O} \cdot 2 \text{ Si} \text{ O}_2 \cdot 20 \text{ WO}_3 \cdot \text{ aq.}$ 

This salt is obtained by careful decomposition of a 12-tungstic acid-1-silicate with  $K_2CO_3$ .

A solution of 8 g. of 2  $K_2O \cdot SiO_2 \cdot 12 WO_3 \cdot aq$ . in a minimum of  $H_2O$  is prepared at room temperature and treated carefully (no heating) with a fairly conc. solution of  $K_2CO_3$  in  $H_2O$  (2 moles of  $K_2CO_3 = 276.4$  g. per mole of 2  $K_2O \cdot SiO_2 \cdot 12 WO_3$ ; one mole of the compounds containing 18 or 9 moles of  $H_2O/mole$  weight 3355.8 or 3193.6 g., respectively). At this point, the solution gives a neutral reaction, and the desired potassium salt crystallizes out immediately with no need for further concentration. The salt is washed with some cold water. Yield: 4.5 g.

PROPERTIES:

Formula weight 5416.78. Sparkling crystals. Water content: 23 moles of  $H_2O/mole$ .

REFERENCE:

F. Kehrmann. Z. anorg. Chem. <u>39</u>, 103 (1904). Checked by the present authors.

#### 12-Tungstic Acid-1-Phosphates

In keeping with their molecular weight and constitution, these compounds should be regarded as derivatives of a 12-tungstic-1-phosphoric acid  $H_3[PO_4(W_3O_9)_4 \cdot aq.]$  [J. F. Keggin, Proc. Roy. Soc. A <u>144</u>, 75 (1934)].

The Sodium Salt 
$$3Na_2O \cdot P_2O_5 \cdot 24WO_3 \cdot aq$$
.  
 $24 Na_2WO_4 (aq.) + 2 Na_2HPO_4 + 46 HCl = 3 Na_2O \cdot P_2O_5 \cdot 24 WO_3 (aq.)$   
 $7053.8 284.0 1677.6 5894.0 + 46 NaCl$   
 $2688.7$ 

A solution of 50 g. of  $Na_2WO_4 \cdot 2 H_2O$  and 25 g. of  $Na_2HPO_4 \cdot 12 H_2O$  in 80 ml. of  $H_2O$  is evaporated until a surface skin of crystals forms; then 75 ml. of 24% HCl (d 1.12) is added with stirring. A precipitate forms momentarily, but then redissolves completely. The solution is reevaporated on a steam bath until a crystal skin begins to form. The product is recrystallized from  $H_2O$ .

PROPERTIES:

Large colorless (sometimes slightly greenish) columnar crystals. Water content: 30 moles of  $H_2O/mole$ . Another hydrate also exists. Soluble in  $H_2O$ .

**REFERENCE:** 

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 251 (1917). Checked by the present author.

The Free Acid  $P_2O_5 \cdot 24WO_3 \cdot aq$ .

This salt is obtained by Drechsel's method (p. 1700), that is, by extracting a solution of 3  $\operatorname{Na}_2O \cdot \operatorname{P}_2O_5 \cdot 24$  WO<sub>3</sub> · aq. with ether and conc. HCl. Light yellow or greenish crystals precipitate. However, if the starting sodium salt is first recrystallized once or twice, the product consists of transparent, colorless crystals.

This acid is also prepared very readily by ion exchange (see p. 1701). The starting solution contains 20 g. of  $3 \text{ Na}_2\text{O} \cdot \text{P}_2\text{O}_5 \cdot 24 \text{ WO}_3 \cdot \text{aq}$ . in 100 ml. of H<sub>2</sub>O. The colorless eluate is concentrated in a vacuum desiccator. Yield: 11 g.

PROPERTIES:

Formula weight 5706.59. The colored crystals disintegrate, often even in a few hours, to a crystalline powder; the colorless

crystals may often be kept for months. Large, lustrous octahedra, soluble in  $H_2O$ . Water content: 63 moles of  $H_2O/mole$ . This hydrate readily converts to a hydrate with 51 moles of  $H_2O/mole$ , which forms trigonal crystals and begins to melt at 89°C. It loses its water of crystallization in a vacuum desiccator over  $H_2SO_4$ .

REFERENCES:

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 251 (1917); G. Jander and D. Ertel, unpublished experiments.

The Potassium Salt  $3K_2O \cdot P_2O_5 \cdot 24WO_3 \cdot aq$ .

The Ammonium Salt  $3(NH_4)_2 \mathbf{0} \cdot \mathbf{P}_2 \mathbf{0}_5 \cdot 24 W \mathbf{0}_3 \cdot \mathbf{a}_4$ .

 $\frac{3 \text{ Na}_2 \text{O} \cdot \text{P}_2 \text{O}_5 \cdot 24 \text{ WO}_3 \text{ (aq.)} + \begin{array}{c} 6 \text{ KCl } [6 \text{ NH}_4 \text{Cl}] \\ 5894.0 & 447.3 & 321.0 \end{array}$ 

 $= 3 K_2 O \cdot P_2 O_5 \cdot 24 WO_3 (aq.) [3 (NH_4)_2 O \cdot P_2 O_5 \cdot 24 WO_3 (aq.)] + 6 NaCI$ 5990.6 350.7 350.7

A solution of the free acid  $P_2O_5 \cdot 24 WO_3 \cdot aq$ . or the sodium salt  $3 Na_2O \cdot P_2O_5 \cdot 24 WO_3 \cdot aq$ . is treated with KCl or NH<sub>4</sub>Cl. A thick white precipitate forms, even if the solutions are very dilute and contain free mineral acids.

PROPERTIES:

Microcrystalline white precipitates, which filter with difficulty. Very poor solubility in  $H_2O$ .

REFERENCE:

F. Kehrmann and M. Freinkel. Ber. dtsch. chem. Ges. <u>24</u>, 2326 (1891). Checked by the present authors.

The Barium Salt  $3 Ba O \cdot P_2 O_5 \cdot 24 WO_3 \cdot aq$ .

$$\begin{array}{c} 3 \text{ Na}_{2}\text{O} \cdot \text{P}_{2}\text{O}_{5} \cdot 24 \text{ WO}_{3} \text{ (aq.)} + 3 \text{ BaCl}_{2} = 3 \text{ BaO} \cdot \text{P}_{2}\text{O}_{5} \cdot 24 \text{ WO}_{3} \text{ (aq.)} + 6 \text{ NaCl} \\ \\ 5894.0 \qquad \qquad 624.8 \qquad 6168.1 \qquad 350.7 \end{array}$$

A saturated solution of 14 g. of  $3 \operatorname{Na_2O} \cdot \operatorname{P_2O_5} \cdot 24 \operatorname{WO_3} \cdot \operatorname{aq}$ . is mixed, while hot, with 60 ml. of hot, saturated BaCl<sub>2</sub> solution. The liquid becomes cloudy, and a heavy, white crystalline precipitate forms on cooling. Concentration of the mother liquor (not too far) gives a second fraction of the desired barium salt. Yield: about 7 g. PROPERTIES:

Well-formed, regular, colorless octahedra, which effloresce in air. Water content: 58 moles of  $\rm H_2O/mole$ . Moderately soluble in  $\rm H_2O$ .

REFERENCE:

F. Kehrmann and M. Freinkel. Ber. dtsch. chem. Ges. <u>24</u>, 2326 (1891). Checked by the present authors.

# 22-Tungstic Acid-2-Phosphates

The Potassium Salt  $7K_2O \cdot P_2O_5 \cdot 22WO_3 \cdot aq$ .

Produced by careful decomposition of a 12-tungstic acid-1-phosphate with  $K_2CO_3$ .

An approximately 30% suspension of  $3 \text{ K}_2 \text{O} \cdot \text{P}_2 \text{O}_5 \cdot 24 \text{ WO}_3 \cdot \text{aq.}$  in water is heated to boiling and treated with about 10%  $\text{K}_2 \text{CO}_3$  solution until solution is complete. Excess  $\text{K}_2 \text{CO}_3$  should be avoided. The solution, which then has a neutral reaction, is evaporated on the steam bath. On cooling,  $7 \text{ K}_2 \text{O} \cdot \text{P}_2 \text{O}_5 \cdot 22 \text{ WO}_3 \cdot \text{aq.}$  separates out. It may be recrystallized from  $\text{H}_2 \text{O}$  which contains some acetic acid.

PROPERTIES:

Formula weight 5902.3. Large, octahedral crystals, partially present as spearlike aggregates, accompanied by a fine powder. Soluble in hot water, less so in cold. Decomposes in the presence of free mineral acid (see next preparation).

REFERENCES:

F. Kehrmann. Z. anorg. Chem. <u>1</u>, 435 (1892); P. Souchay. Ann. Chimie [12] <u>2</u>, 204 (1947). Checked by the present authors.

## 21-Tungstic Acid-2-Phosphates

The Potassium Salt  $3K_2O \cdot P_2O_5 \cdot 21WO_3 \cdot aq$ .

Produced from  $7 K_2 O \cdot P_2 O_5 \cdot 22 WO_3 \cdot aq$ . by treatment with HCl. In addition to the desired compound, the potassium salt of 12-tungstic-1-phosphoric acid,  $3 K_2 O \cdot P_2 O_5 \cdot 24 WO_3 \cdot aq$ ., is also formed.

A conc. solution of  $7 \text{ K}_2\text{O} \cdot \text{P}_2\text{O}_5 \cdot 22 \text{ WO}_3 \cdot \text{aq.}$  is prepared at the boiling point, and dilute (about 7%) HCl is added dropwise until the solution becomes acidic. An insoluble white precipitate of  $3 \text{ K}_2\text{O} \cdot \text{P}_2\text{O}_5 \cdot 24 \text{ WO}_3 \cdot \text{aq.}$  appears. This is filtered off, and the filtrate is treated with KCl powder to salt out the desired compound. The latter is recrystallized from H<sub>2</sub>O containing two drops of HCl.

#### PROPERTIES:

Formula weight 5293.6. Relatively large, lustrous, hexagonal columns, partly intergrown. Readily soluble in  $H_2O$ . The aqueous solution is unstable on boiling, depositing a white precipitate.

**REFERENCES:** 

F. Kehrmann. Z. anorg. Chem. <u>1</u>, 436 (1892); P. Souchay. Ann. Chimie [12] 2, 204 (1947). Checked by the present authors.

# 18-Tungstic Acid-2-Phosphates

Based on molecular weight determinations, these compounds should be regarded as salts of an 18-tungstic-2-phosphoric acid  $H_{6}[(PO_{4})_{2}(W_{3}O_{9})_{6} \cdot aq.]$  [G. Jander and F. Exner, Z. phys. Chem. (A) <u>190</u>, 195 (1942)].

The Ammonium Salt  $3(NH_4)_2 O \cdot P_2 O_5 \cdot 18 WO_3 \cdot aq$ .

A solution of  $Na_2WO_4 \cdot aq$ . is boiled for a long time with a large excess of phosphoric acid. This involves an apparently slow condensation reaction. The resulting solution of the sodium salt of 18-tungstic-1-phosphoric acid is treated with solid  $NH_4Cl$  to salt out the ammonium salt.

One mole of  $Na_2WO_4 \cdot 2 H_2O$  is dissolved in hot  $H_2O$  and treated with four moles of phosphoric acid (in the form of a conc. solution, d 1.17) and about 100 ml. of additional  $H_2O$ . The yellow solution is boiled for 3-5 hours while stirring and replacing the water lost by evaporation; the boiling point is 108°C. To remove any reduction products which may have formed, a few drops of nitric acid are added at the end. As the solution cools, solid NH<sub>4</sub>Cl is added until the desired ammonium salt 3 (NH<sub>4</sub>)<sub>2</sub>O · P<sub>2</sub>O<sub>5</sub> · 18 WO<sub>3</sub> is completely precipitated and the solution becomes colorless. The salt is filtered out, redissolved in hot  $H_2O$ , and reprecipitated with conc. NH<sub>4</sub>Cl solution. It is then filtered out again, washed and recrystallized twice from water; the first crystal fraction is discarded each time. An analytically pure preparation is thus obtained. SYNONYM:

Ammonium luteophosphotungstate.

**PROPERTIES:** 

Formula weight 4471.7. Lemon-yellow or pale yellow-green triclinic crystals. Water content: 14 moles of  $H_2O/mole$ . Soluble in  $H_2O$ .

REFERENCES:

F. Kehrmann. Z. anorg. Chem. <u>1</u>, 432 (1892); Ber. dtsch. chem. Ges. <u>20</u>, 1808 (1887); G. Jander and H. Banthien. Z. anorg. allg. Chem. <u>229</u>, 142 (1936).

The Free Acid  $P_2O_5 \cdot 18WO_3 \cdot aq$ .

Obtained by the method of Drechsel (see p. 1700 f.) from the solution of the sodium salt described in the previous preparation. It can also be prepared by ion exchange (cf. p. 1701), using a

solution of 10 g. of 3  $(NH_4)_2 O \cdot P_2 O_5 \cdot 18 WO_3 \cdot aq$ . in 50 ml. of  $H_2O$ . The eluate is clear and pale yellow-green. It is concentrated in a vacuum desiccator. Yield: 9 g.

SYNONYM:

Luteophosphotungstic acid.

PROPERTIES:

Formula weight 4315.4. M.p. 28°C. Lemon-yellow hexagonal tablets. Water content: 42 moles of  $H_2O/mole$ . Readily soluble in  $H_2O$ .

**REFERENCES:** 

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 261 (1917); G. Jander and F. Exner. Z. phys. Chem. (A) <u>190</u>, 195 (1942); G. Jander and D. Ertel. Unpublished experiments.

## 12-Tungstic Acid-1-Arsenates

By analogy with the 12-tungstic acid-1-phosphates, compounds of this class should be regarded as salts of a 12-tungstic-1arsenic acid  $H_3[AsO_4(W_3O_9)_4 \cdot aq.]$  [J. W. Illingworth and J. F. Keggin, J. Chem. Soc. (London) 1935, 575]. All tungstic acid arsenates resemble closely the tungstic acid phosphates in their manner of preparation and their behavior. They are only a little less stable.

The Ammonium Salt  $3(NH_4)_2O \cdot As_2O_5 \cdot 24WO_3 \cdot aq$ .

Obtained by addition of NH<sub>4</sub>Cl to a solution of the corresponding sodium salt  $3 \text{ Na}_2\text{O} \cdot \text{As}_2\text{O}_5 \cdot 24 \text{ WO}_3 \cdot \text{aq.}$ , which is not known to exist as a solid.

$$3 \operatorname{Na_2O} \cdot \operatorname{As_2O_5} \cdot 24 \operatorname{WO_3} (\operatorname{aq.}) + 6 \operatorname{NH_4Cl}_{321.0} = 3 (\operatorname{NH_4}_2 \operatorname{O} \cdot \operatorname{As_2O_5} \cdot 24 \operatorname{WO_3} (\operatorname{aq.}) + 6 \operatorname{NaCl}_{5952.1}$$

A solution of 52.8 g. of  $Na_2WO_4 \cdot 2 H_2O$  is prepared by heating a mixture of the salt and sufficient water to make the final volume 90 ml. A second solution is prepared from 2.3 g. of  $As_2O_5$  and 15 ml. of very concentrated aqueous NaOH, and is then diluted with water to a final volume of 70 ml. After cooling, both solutions are combined and treated with conc. HCl until the mixture is strongly acid (pH paper). This requires 15-20 ml. of conc. HCl. The resulting mixture is unstable; on long standing, a white sediment is formed. Therefore 21 g. of solid NH<sub>4</sub>Cl is added immediately; the mixture is heated once to boiling and allowed to stand for two hours on a steam bath. The white precipitate is filtered out, washed first with NH<sub>4</sub>Cl solution acidified with HCl and then with some cold H<sub>2</sub>O, and dried in a desiccator.

PROPERTIES:

Fine white, crystalline precipitate. Water content: 12 moles of  $H_2O/mole$ . Relatively sparingly soluble in  $H_2O$ .

**REFERENCES:** 

F. Kehrmann. Z. anorg. Chem. <u>22</u>, 286 (1900); A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 268 (1917). Checked by the present authors.

## 18-Tungstic Acid-2-Arsenates

These compounds are exactly the same in appearance and water content as the 18-tungstic acid-2-phosphates. The ammonium salt and the free acid are obtained by methods used for those compounds (see p. 1723), using arsenic acid instead of phosphoric. Formula weights:  $As_2O_5 \cdot 18 WO_3 \cdot aq_{.:} 4404.4; 3 (NH_4)_2 O \cdot As_2O_5 \cdot 18 WO_3 \cdot aq_{.:} 4560.6.$ 

**REFERENCES:** 

F. Kehrmann. Z. anorg. Chem. <u>22</u>, 290 (1900); A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. 101, 270 (1917).

#### 6-Tungstic Acid-1-Tellurates

In keeping with their molecular weight, these compounds should be considered 6-tungstic acid-1-tellurates (salts of a 6-tungstic-1-telluric acid  $H_6[TeO_6 \cdot 6 WO_3 \cdot aq.])$  [G. Jander and K.F. Jahr, Kolloid-Beihefte <u>41</u>, 308 (1935)].

The Guanidinium Salt  $3(CN_3H_6)_2 O \cdot TeO_3 \cdot 6WO_3 \cdot aq$ .

Tungsten (VI) oxide is dissolved in guanidinium carbonate, HCl is added to ensure the level of acidity required for the formation of a heteropoly compound, and the required quantity of telluric acid is introduced.

$$\begin{array}{rl} 6\,(\mathrm{CN_3H_6})_2\mathrm{CO_3} &+ \,\mathrm{H_6TeO_6} &+ \,6\,\mathrm{WO_3} &+ \,6\,\mathrm{HCl} \\ 1081.0 & 229.6 & 1391.5 & 218.8 \\ &= \,3\,(\mathrm{CN_3H_6})_2\mathrm{O}\cdot\mathrm{TeO_3}\cdot6\,\mathrm{WO_3}\,(\mathrm{aq.}) &+ \,6\,(\mathrm{CN_3H_6})\mathrm{Cl} \,+ \,6\,\mathrm{CO_2} \\ & 1975.6 & 513.1 \end{array}$$

A boiling aqueous solution of 0.06 moles of guanidinium carbonate is gradually treated (stirring) with 0.06 moles of fine yellow tungstic acid powder (not too strongly ignited). The tungstic acid dissolves. The solution is filtered, and 0.06 moles of HCl is added to the clear filtrate. The nascent precipitate is redissolved by addition of hot  $H_2O$ . Then, 0.01 moles of telluric acid is added. On cooling, the desired salt crystallizes out. It is recrystallized from hot  $H_2O$ .

**PROPERTIES:** 

Pure white, well-formed platelike crystals. Relatively poor solubility in  $H_2O$ . Water content: 3 moles of  $H_2O$ /mole.

REFERENCE:

R. Häberle. Thesis, Univ. of Berlin, 1911.

1726

#### Metatungstates, Dodecatungstates

In keeping with their molecular weight and structure, these compounds should be considered salts of a dodecatungstic acid  $H_6[H_2O_4(W_3O_9)_4 \cdot aq.]$  [see, for example, G. Jander, Z. phys. Chem. (A) <u>187</u>, 149 (1940); R. Signer and H. Gross, Helv. Chim. Acta <u>17</u>, 1076 (1934); G. Schott and C. Harzdorf, Z. anorg. allg. Chem. <u>288</u>, 15 (1956)]. P. Souchay [Ann. Chim. (11) <u>18</u>, 1; 169 (1943)] terms the product obtained by acidification of a monotungstate a  $\psi$ -metatungstate. These compounds are hexatungstates, and also differ chemically from the "true" metatungstates, but they are not identical with the "paratungstates." [See G. Jander and U. Krüerke, Z. anorg. allg. Chem. <u>265</u>, 244 (1951).]

The Sodium Salt  $Na_2O \cdot 4WO_3 \cdot aq$ .

I. 
$$5 \operatorname{Na_2O} \cdot 12 \operatorname{WO_3} (\operatorname{aq.}) + 8 \operatorname{WO_3} = 5 \left[ \operatorname{Na_2O} \cdot 4 \operatorname{WO_3} (\operatorname{aq.}) \right]_{3092.2}$$
 1854.9 4947.1

A dilute solution of "sodium paratungstate"  $5 \text{ Na}_2 \text{O} \cdot 12 \text{ WO}_3 \cdot \text{aq.}$  is boiled with an excess of yellow tungstic acid hydrate until a filtered sample no longer gives a precipitate of tungstic acid when treated with dil. HCl. The solution is then filtered to remove the excess tungstic acid and the insoluble white products formed during boiling. The filtrate is concentrated somewhat on a steam bath and allowed to crystallize in a desiccator over H<sub>2</sub>SO<sub>4</sub>.

II. The salt may be prepared more simply as follows:

$$\begin{array}{rl} Na_2WO_4\,(aq.) \ + \ 3\,WO_3 \ = \ Na_2O \cdot 4\,WO_3\,(aq.) \\ 293.8 & 695.6 & 989.4 \end{array}$$

A solution of 20 g. of  $Na_2WO_4 \cdot 2 H_2O$  in 200 ml. of  $H_2O$  is prepared and an excess of yellow tungstic acid is added to it in portions. The suspension is boiled for about 1.5 hours, which produces white insoluble precipitates, settling out together with the excess tungstic acid. The pH of the solution after boiling and filtration is about 3. It is concentrated as described above.

SYNONYM:

Sodium metatungstate.

PROPERTIES:

Colorless tetragonal bipyramids. Water content: 10 moles of  $H_2O/mole$ . The crystals effloresce easily, and lose almost all their water over  $H_2SO_4$ . Readily soluble in  $H_2O$ .

**REFERENCES:** 

I. C. Scheibler. J. prakt. Chem. <u>83</u>, 301 (1861).

II. Authors' experiments.

The Silver Salt  $Ag_2 O \cdot 4WO_3 \cdot aq$ .

 $\begin{array}{rl} Na_{2}O \cdot 4 WO_{3} (aq.) &+ 2 AgNO_{3} = Ag_{2}O \cdot 4 WO_{3} (aq.) &+ 2 NaNO_{3} \\ & 989.4 & 339.8 & 1159.2 & 170.0 \end{array}$ 

A solution of "sodium metatungstate"  $Na_2O \cdot 4 WO_3 \cdot aq$ . is allowed to react with a solution containing the equivalent quantity of AgNO<sub>3</sub>. A white precipitate slowly crystallizes out.

**PROPERTIES:** 

Small white scales. Water content: 3 moles of  $H_2O/mole$ . Quite insoluble in  $H_2O$ .

**REFERENCE:** 

A. Rosenheim and F. Kohn. Z. anorg. Chem. <u>69</u>, 250 (1911).

The Free Acid  $H_2O \cdot 4WO_3 \cdot aq$ .

The free acid may be isolated from a solution of  $Na_2O \cdot 4 WO_3 \cdot$ aq. (see above for preparation) by Drechsel's method (see p. 1700 f.). The oily ether addition product is best distributed onto several watch glasses, which are then placed in a fast stream of dry air. The product is then rapidly dried by pressing on clay plates. This affords a relatively stable preparation, which is soluble in H<sub>2</sub>O, forming a clear solution. Some preparations convert to yellow tungstic acid in only a few days.

Better results are obtained if the acid is prepared by ion exchange (see p. 1701) from crystalline "sodium metatungstate." A solution of 20 g. of the sodium salt  $Na_2O \cdot 4 WO_3 \cdot aq$ . in 50 ml. of  $H_2O$  is used. The clear eluate does not hydrolyze when concentrated in a vacuum desiccator. White crystals. Yield: 18 g.

PROPERTIES:

Formula weight 945.5. Large octahedra; according to some authors, also rhombohedra or bipyramids; readily effloresce in air. Water content: 8 moles of  $H_2O/mole$ ; some authors report deviation from this value. Readily soluble in  $H_2O$ . Dilute solutions may be kept in the cold for extended periods of time, but coagulation occurs on heating. Concentrated solutions often coagulate

1728

even at moderately high temperatures. Because of its constitution and chemical behavior, this acid should be considered a heteropoly compound.

REFERENCES

A. Rosenheim and F. Kohn. Z. anorg. Chem. <u>69</u>, 253 (1911); G. Jander and D. Ertel. Unpublished experiments.

# 12-Molybdic Acid-1-Silicates

In keeping with their constitution, all compounds of this composition are salts of a 12-molybdic-1-silicic acid H<sub>4</sub>[SiO<sub>4</sub> (Mo<sub>3</sub>O<sub>9</sub>)<sub>4</sub> · aq.] [J. W. Illingworth and J. F. Keggin, J. Chem. Soc. (London) <u>1935</u>, 575].

The Sodium Salt  $2Na_2O \cdot SiO_2 \cdot 12MoO_3 \cdot aq$ .

The procedure given below affords a nitric acid-soluble solution of the sodium salt, which is the starting material for the preparation of the rubidium or cesium salts. Since this solution of the sodium salt also contains a very large quantity of  $NaNO_3$ , it cannot be used directly for the preparation of the crystalline sodium salt or the free acid (by Drechsel's method).

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\begin{array}{rl} 12 \operatorname{Na_2MoO_4}\left(\operatorname{aq.}\right) \ + \ \operatorname{Na_2SiO_3} \ + \ 22 \ \mathrm{HNO_3} \\ & & \\ 2471.2 & 122.1 \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & & \\ & &
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Solid NaOH (60 g.) is dissolved in 400 ml. of boiling  $H_2O$  and a total of 172g, of MoO<sub>3</sub> (free of ammonium salt) is added in portions during the course of 10-15 min. When this is completely dissolved, boiling is interrupted and 500 ml. of cold  $H_2O$  is poured in. Then 250 ml. of HNO<sub>3</sub> (d 1.39) is made up to 350 ml. with water, and portions of this are rapidly added to the molybdate with constant stirring. No appreciable amount of precipitate should form. The sodium silicate solution described below is added immediately in a thin stream and with constant stirring.

The sodium silicate solution is made by dissolving 28 g, of commercial crystalline  $Na_2SiO_3 \cdot 9 H_2O$  in 125 ml. of 2N NaOH and boiling 10-15 minutes to effect conversion to the monosilicate.

The solution of 2  $Na_2O \cdot SiO_2 \cdot 12 MoO_3$  is intensely yellow and is not as stable as solutions of other heteropolysalts. It is therefore advisable to maintain the conditions specified above, particularly as far as the H<sup>+</sup> concentration is concerned. A solution of the sodium salt is used in the potassium industry for recovering rubidium and cesium from carnallite. Formula weight 1911.5.

**REFERENCE**:

G. Jander and F. Busch. Z. anorg. allg. Chem. <u>187</u>, 173 (1930).

The Rubidium Salt  $2 \operatorname{Rb}_2 O \cdot \operatorname{Si} O_2 \cdot 12 \operatorname{Mo} O_3 \cdot \operatorname{aq}$ .

The Cesium Salt  $2 \operatorname{Cs}_2 0 \cdot \operatorname{Si} 0_2 \cdot 12 \operatorname{Mo} 0_3 \cdot \operatorname{aq}$ .

$$\begin{array}{rl} 2 \operatorname{Na_2O} \cdot \operatorname{SiO_2} \cdot 12 \operatorname{MoO_3}\left(\operatorname{aq.}\right) &+ 4 \operatorname{RbCl} \left[ 4 \operatorname{CsCl} \right] \\ & 1911.5 & 483.8 & 673.5 \\ &= 2 \operatorname{Rb_2O} \cdot \operatorname{SiO_2} \cdot 12 \operatorname{MoO_3}\left(\operatorname{aq.}\right) \left[ 2 \operatorname{Cs_2O} \cdot \operatorname{SiO_2} \cdot 12 \operatorname{MoO_3}\left(\operatorname{aq.}\right) \right] &+ 4 \operatorname{NaCl} \\ & 2161.4 & 2351.1 & 233.8 \end{array}$$

A nitric acid solution of the sodium salt described above is treated at about  $65^{\circ}$ C with a solution of RbCl or CsCl. Cooling to  $40-50^{\circ}$ C gives a fine yellow crystalline precipitate of the rubidium or cesium salt.

**PROPERTIES:** 

Fine yellow powder. Relatively poor solubility in cold  $H_2O$ , better in hot.

**REFERENCE**:

G. Jander and H. Faber. Z. anorg. allg. Chem. 179, 323 (1929).

## 12-Molybdic Acid-1-Phosphates

In keeping with their structure, compounds of this class should be classified as salts of 12-molybdic-1-phosphoric acid  $H_3[PO_4(Mo_3O_9)_4 \cdot aq.]$  [J. W. Illingworth and J. F. Keggin, J. Chem. Soc. (London) <u>1935</u>, 575].

The Ammonium Salt  $3(NH_4)_2 O \cdot P_2 O_5 \cdot 24 M_0 O_3 \cdot aq$ .

$$\begin{array}{rl} 2 \left[ 5 \left( \mathrm{NH_4} \right)_2 \mathrm{O} \cdot 12 \operatorname{MoO_3}\left( \mathrm{aq.} \right) \right] &+ 2 \operatorname{Na_2HPO_4} &+ 18 \operatorname{HNO_3} \\ & & & & & \\ 3975.6 & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & & \\ & & & & & \\ & & & & & \\ & & & & & \\ & & & & & & \\ & &$$

A solution of 50 g. of  $Na_2HPO_4$  in a mixture of 300 ml. of nitric acid (d 1.48) and 300 ml. of  $H_2O$  is prepared, cooled and

mixed with a clear, cold solution of 200 g. of commercial ammonium molybdate [this generally is 5  $(NH_4)_2 O \cdot 12 MoO_3 \cdot aq.;$  see p. 1711] in the minimum of  $H_2O$ ; the last solution is added in a thin stream and with stirring. A precipitate forms immediately. It is washed with hot  $H_2O$  to which a few drops of conc.  $HNO_3$  have been added.

The ammonium salt is of value in analytical chemistry, where it is used for the determination and separation of phosphoric acid.

**PROPERTIES**:

Deep yellow, microcrystalline salt, soluble with great difficulty.

**REFERENCE:** 

F. Kehrmann. Z. anorg. Chem. 7, 417 (1894).

The Free Acid  $P_2O_5 \cdot 24MoO_3 \cdot aq$ .

 $\begin{array}{rrrr} 2 \ H_{3} PO_{4} &+& 24 \ MoO_{3} &=& P_{2}O_{5} \cdot 24 \ MoO_{3} \ (aq.) \\ 196.0 & & 3454.8 & & 3596.8 \end{array}$ 

About 35 g. of  $MoO_3$  is added in portions to a boiling solution of 6.3 g. of 25% phosphoric acid in 100 ml. of  $H_2O$ , and the boiling is continued for another 2-2.5 hours. The insolubles are removed by filtration and the yellow solution is shaken with ether to purify the crude product. It is unnecessary to add acid here. Further workup is the same as on p. 1701. Yield: about 20 g. The crystals thus obtained may be recrystallized from a small amount of hot water to which some HNO<sub>3</sub> has been added.

The success of the preparation depends on the availability of ammonium-free  $MoO_3$ . Commerical  $MoO_3$  frequently contains some ammonium ions. To purify this material, the proper quantity is dissolved in an excess of pure aqueous NaOH and the solution is boiled until  $NH_3$  can no longer be detected. The molybdic acid is reprecipitated by careful addition of conc.  $HNO_3$ . It is freed of  $HNO_3$  and  $NaNO_3$  by several decantantions with water and filtered out.

PROPERTIES:

Well-formed orange-yellow octahedra. Very readily soluble in  $H_2O$ . Water content: 63 moles of  $H_2O$ /mole. Melting range 78 to 98°C. Other hydrates also exist. REFERENCE:

A. Rosenheim and J. Jaenicke. Z. anorg. allg. Chem. <u>101</u>, 248 (1917). Checked by the present authors.

The Barium Salt  $3 BaO \cdot P_2 O_5 \cdot 24 MoO_3 \cdot aq$ .

This salt is obtained by treatment of the free acid with BaCl<sub>2</sub> solution.

$$\begin{array}{rl} P_{2}O_{5} \cdot 24 \text{ MoO}_{3}(\text{aq.}) &+ 3 \text{ BaCl}_{2} &+ 3 \text{ H}_{2}O \\ && & & & & \\ 3596.8 & & & & 624.9 & & 54.0 \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & \\ && & & & \\ && & & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & & \\ && & & & \\ && & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & & \\ && & & & & \\ && & & & & & \\ && & & & &$$

A clear conc. solution of the free acid is mixed with excess hot, saturated  $BaCl_2$  solution. The barium salt separates at once as coarse crystals. These are filtered out and washed with small amounts of cold  $H_2O$ , then recrystallized twice from hot  $H_2O$  to which some HNO<sub>3</sub> has been added.

PROPERTIES:

Lemon-yellow octahedra; appreciably soluble in H<sub>2</sub>O.

REFERENCE:

F. Kehrmann. Z. anorg. Chem. <u>7</u>, 417 (1894). Checked by the present authors.

#### 18-Molybdic Acid-2-Phosphates

By analogy with the 18-tungstic acid-2-phosphates, compounds of this class should be classified as salts of an 18-molybdic-2phosphoric acid  $H_6[(PO_4)_2(Mo_3O_9)_6 \cdot aq_.]$ .

The Free Acid  $P_2O_5 \cdot 18MoO_3 \cdot aq$ .

The sodium salt of 18-molybdic-2-phosphoric acid is prepared first.

 $2 \text{ Na}_{3}\text{PO}_{4} + 18 \text{ MoO}_{3} = 3 \text{ Na}_{2}\text{O} \cdot \text{P}_{2}\text{O}_{5} \cdot 18 \text{ MoO}_{3} \text{ (aq.)}$ 327.9 2591.1 2919.0

The free acid is then obtained by Drechsel's method. Since the resulting product is still contaminated with 12-molybdic-1-phosphoric acid, the pure 12-molybdic-1-phosphoric acid is prepared by addition of  $H_3PO_4$ .

A boiling solution of Na<sub>3</sub>PO<sub>4</sub> is treated with portions of MoO<sub>3</sub>  $(Na_3PO_4:MoO_3 \text{ mole ratio} = 1:9)$ . The solution is filtered and concentrated to a small volume. Any reduction products present are oxidized by means of some bromine water. The free acid is obtained from the solution by extraction with ether and HCl (see p. 1700 f.). The 18-molybdic-2-phosphoric acid thus prepared is never pure, but still contains rather large amounts of 12molybdic-1-phosphoric acid. Therefore the aqueous solution of the acid is treated with sirupy phosphoric acid, the latter being added in a quantity corresponding to the amount missing in the formula. (Roughly, it may be assumed that about 1/3 of the heteropolyacid product is still in the form of 12-molybdic-1phosphoric acid. To convert this, two moles of H<sub>3</sub>PO<sub>4</sub> are required for three moles of  $P_2O_5 \cdot 24$  MoO<sub>3</sub> · aq.) The aqueous solution is allowed to stand until NH<sub>4</sub> or K salts cause no further precipitation. The free 18-molybdic-2-phosphoric acid then separates out in a vacuum desiccator over conc. H<sub>2</sub>SO<sub>4</sub>.

SYNONYM:

Luteophosphomolybdic acid.

**PROPERTIES:** 

Formula weight 2733.06. Orange-colored prisms, readily soluble in  $H_2O$ . Several hydrates exist; various water contents are reported by individual authors. Quite unstable; an aqueous solution soon reverts to 12-molybdic-1-phosphoric acid.

REFERENCE:

G. Jander and E. Drews. Z. phys. Chem. (A) 190, 228 (1942).

The Potassium Salt  $3K_2O \cdot P_2O_5 \cdot 18M_0O_3 \cdot aq$ .

May be obtained from a solution of the free acid by salting out with solid KCl.

A solution of 18-molybdic-2-phosphoric acid, as concentrated as possible, is treated in the cold with KCl powder until the desired potassium salt separates as a yellow precipitate.

PROPERTIES:

Formula weight 3015.64. Orange yellow, prismatic crystals. Water content: 14 moles of  $H_2O/mole$ . Soluble in  $H_2O$ .

**REFERENCE:** 

F. Kehrmann. Z. anorg. Chem. <u>7</u>, 147 (1894). Checked by the present authors.

## 12-Molybdic Acid-1-Arsenates

By analogy to the 12-molybdic acid-1-phosphates, which they very closely resemble, all compounds of this class should be classified as salts of a 12-molybdic-1-arsenic acid  $H_3[AsO_4(Mo_3O_9)_4 \cdot aq.]$ .

The Potassium Salt  $3K_2O \cdot As_2O_5 \cdot 24M_0O_3 \cdot aq$ .

Commercial ammonium molybdate 5  $(NH_4)_2 O \cdot 12 MoO_3 \cdot aq$ . (see p. 1711) is heated with an excess of aqueous KOH, forming a solution of  $K_2MoO_4 \cdot aq$ . Addition of HNO<sub>3</sub> and  $As_2O_3$  yields the desired salt.

Commercial ammonium molybdate (30 g.) is heated in a porcelain dish with an aqueous solution of KOH (one part of KOH by weight to two parts of  $H_2O$ ) until all  $NH_3$  has been driven off. (An excess of KOH should be avoided, since excessive quantities of KNO<sub>3</sub> are then formed during the reaction; this can crystallize out under some circumstances and contaminate the product.) After cooling, the solution is diluted with 50 ml of  $H_2O$  and slowly poured into an excess of conc.  $HNO_3$ ; external cooling may be used if required. The solution remains clear. It becomes deep yellow on addition of the stoichiometric quantity of  $As_2O_5$  in 50 ml. of  $H_2O$ . It is briefly heated to 60-70°C to produce a yellow precipitate, more of which appears on cooling. Yield: about 8 g.

PROPERTIES:

Fine, yellow crystalline powder. Water content: 12 moles of  $H_2O/mole$ . Not particularly soluble in cold water, somewhat more soluble in hot.

REFERENCE:

O. Pufahl. Thesis, Univ. of Leipzig, 1888. Checked by the present authors.

# 18-Molybdic Acid-2-Arsenates

By analogy with the 18-molybdic acid-2-phosphates, these compounds should probably be classified as salts of an 18-molybdic-2-arsenic acid  $H_6[(ASO_4)_2(Mo_3O_9)_6 \cdot aq.]$ .

The Sodium Salt  $3 \operatorname{Na}_2 0 \cdot \operatorname{As}_2 0_5 \cdot 18 \operatorname{Mo} 0_3 \cdot \operatorname{aq}$ .

Since the 18-molybdic acid-2-arsenates are more stable than the 12-molybdic acid-1-arsenates,  $3 \text{ Na}_2\text{O} \cdot \text{As}_2\text{O}_5 \cdot 18 \text{ MoO}_3 \cdot \text{aq}$ . is produced when a sodium arsenate solution is treated with excess MoO<sub>3</sub>.

> $2 \operatorname{Na_3AsO_4} + 18 \operatorname{MoO_3} = 3 \operatorname{Na_2O} \cdot \operatorname{As_2O_5} \cdot 18 \operatorname{MoO_3} (aq.)$ 415.8 2591.1 3006.9

A solution of  $Na_3AsO_4$  is saturated with  $MoO_3$  at the boil, and boiling is continued for some time. The deep-yellow solution is then filtered and concentrated. The desired sodium salt precipitates as yellow crystals.

**PROPERTIES:** 

Yellow monoclinic crystals. Water content: 23 or 24 moles of  $H_2O/mole$ . Soluble in  $H_2O$ .

REFERENCE:

A. Rosenheim and A. Traube. Z. anorg. allg. Chem. 91, 92 (1915).

The Free Acid  $As_2O_5 \cdot 18MoO_3 \cdot aq$ .

I. Drechsel's method (see p. 1700) (that is, extraction of a solution of the sodium salt with HCl and ether) is used.

II.  $18 \operatorname{BaMoO_4} + \operatorname{As_2O_5} + 18 \operatorname{H_2SO_4} = \operatorname{As_2O_5} \cdot 18 \operatorname{MoO_3}(\operatorname{aq.}) + 18 \operatorname{BaSO_4}$ 5351.6 229.8 1765.4 2820.9 4201.6

Preparation of BaMoO<sub>4</sub>: About 50 g. of commercial ammonium molybdate is dissolved in about 300 ml. of boiling H<sub>2</sub> (some ammonia is added). This solution is introduced gradually into a solution of 100 g. of Ba(OH)<sub>2</sub>  $\cdot$  8 H<sub>2</sub>O in 300 ml. of H<sub>2</sub>O heated on a steam bath. The mixture is heated and stirred for 2-3 hours longer. The white precipitate is washed several times with hot H<sub>2</sub>O, then heated once more with baryta water [aqueous Ba(OH)<sub>2</sub>] and thoroughly washed.

The BaMoO<sub>4</sub> is suspended in a solution of arsenic acid so that one g.-atom of As is present per nine moles of  $MoO_3$ . Then the Ba is precipitated by addition of the stoichiometric quantity of H<sub>2</sub>SO<sub>4</sub> (mechanical stirring). The filtered yellow solution is concentrated in vacuum at about 40°C and crystallized over conc. H<sub>2</sub>SO<sub>4</sub>. PROPERTIES:

Deep-red triclinic crystals; decompose readily when stored damp. Water content: 28 moles of  $H_2O/mole$ . Quite soluble in  $H_2O$ . A yellow hydrate containing more water also exists.

**REFERENCES:** 

A. Rosenheim and A. Traube. Z. anorg. allg. Chem <u>91</u>, 91 (1915);
 O. Pufahl. Thesis, Univ. of Leipzig, 1888.

#### 6-Molybdic Acid-2-Arsenates

According to molecular weight studies, the compounds of this class are salts of a 6-molybdic-2-arsenic acid  $H_6[(ASO_4)_2(Mo_3O_9)_2 \cdot aq.]$  [G. Jander and E. Drews, Z. phys. Chem. (A) <u>190</u>, 219 (1942)].

The Free Acid  $As_2O_5 \cdot 6MoO_3 \cdot aq$ .

A solution of 2.6 g. of  $As_2O_5$  in 200 ml. of boiling  $H_2O$  is prepared; the hot solution is poured onto 20 g. of BaMoO<sub>4</sub> (one g.-atom of As per three moles of MoO<sub>3</sub>). A solution of 6.6 g. of conc.  $H_2SO_4$  in about 20 ml. of  $H_2O$  is added, and the mixture is heated for one hour on a steam bath (stirring). The BaSO<sub>4</sub> precipitate is filtered off, and the clear (sometimes greenish) solution is concentrated in a vacuum desiccator until crystallization.

PROPERTIES:

Large, fragile, colorless crystalline scales. Water content: 18 moles of  $H_2O/mole$ . Soluble in  $H_2O$ . The compound is a strongly held, very stable complex.

REFERENCE:

O. Pufahl. Thesis, Univ. of Leipzig, 1888. Checked by the present authors.

The Sodium Salt  $Na_2O \cdot As_2O_5 \cdot 6MoO_3 \cdot aq$ .

A solution of sodium paramolybdate  $5 \operatorname{Na}_2O \cdot 12 \operatorname{MoO}_3 \cdot \operatorname{aq}$ . is treated with the stoichiometric quantity of As<sub>2</sub>O<sub>5</sub>, and the salt is synthesized by addition of HCl. However, only half of the theoretical quantity of HCl is used, since otherwise the salt decomposes.

A solution of "sodium paramolybdate" is prepared by dissolving three moles (431.9 g.) of  $MoO_3$  in three moles (120.0 g.) of NaOH. One mole of  $H_3AsO_4$  is added, and the clear solution is gradually treated with one mole of HCl. The desired salt crystallizes out on concentrating the solution.

PROPERTIES:

Formula weight 1155.5. Transparent, lustrous prisms. Water content: 12 moles of  $H_2O/mole$ . Readily soluble in  $H_2O$ .

REFERENCE:

A. Rosenheim and A. Traube. Z. anorg. allg. Chem. 91, 88 (1915).

## 12-Molybdic Acid-2-Chromites

The Ammonium Salt  $3(NH_4)_2 O \cdot Cr_2 O_3 \cdot 12 Mo O_3 \cdot aq$ .

$$5 (NH_4)_2 O \cdot 12 MoO_3 (aq.) + 2 KCr(SO_4)_2 (aq.) + H_2O$$

$$\frac{1987.8}{566.5} \frac{566.5}{18.0} = 3 (NH_4)_2 O \cdot Cr_2O_3 \cdot 12 MoO_3 (aq.) + 2 KHSO_4 + 2 (NH_4)_2SO_4$$

$$2035.7 272.3 264.3$$

A solution of 2 g. of  $KCr(SO_4)_2 \cdot 12 H_2O$  in 20 ml. of  $H_2O$  is heated to boiling and a solution of 30 g. of ammonium paramolybdate 5 (NH<sub>4</sub>)<sub>2</sub>O · 12 MoO<sub>3</sub> · aq. in 110 ml. of H<sub>2</sub>O is slowly added. During this operation the color of the solution changes from green to brownish and then bluish-rose. The pH of the solution should be about 5. On cooling, the desired rose-colored salt crystallizes out, although it sometimes takes 24 hours for this to occur. Yield: about 6 g.

PROPERTIES:

Relatively large, rose-colored rectangular platelets or scales. Water content: 20 moles of  $H_2O/mole$ . Quite soluble in hot  $H_2O$ , somewhat poorer solubility in cold  $H_2O$ . A relatively weak complex.

REFERENCE:

A. Rosenheim and H. Schwer. Z. anorg. allg. Chem. <u>89</u>, 226 (1914). Checked by the present authors.

## 6-Molybdic Acid-1-Periodates

These compounds are salts of a 6-molybdic-1-periodic acid  $H_5[IO_6 \cdot 6 MoO_3 \cdot aq.]$  [G. Jander and K. F. Jahr, Kolloid-Beihefte <u>41</u>, 305 (1935)].

The Sodium Salt  $5 \operatorname{Na}_2 \mathbf{0} \cdot \mathbf{I}_2 \mathbf{0}_7 \cdot 12 \operatorname{MoO}_3 \cdot \mathbf{a}_7$ .

 $\begin{array}{rl} 2 \operatorname{Na_2H_3IO_6} + & 12 \operatorname{MoO_3} + & 3 \operatorname{Na_2CO_3} \\ & 543.9 & 1727.4 & 318.0 \\ & & = & 5 \operatorname{Na_2O} \cdot \operatorname{I_2O_7} \cdot & 12 \operatorname{MoO_3}(\operatorname{aq.}) + & 3 \operatorname{CO_2} \\ & & & 2393.6 \end{array}$ 

It has been found advisable to use two moles of  $Na_2CO_3$  rather than the three moles called for by the equation.

A mixture of 10 parts by weight of  $Na_2H_3IO_6$  and almost 32 parts of  $MoO_3$  in about 120 parts of  $H_2O$  is heated. After a short time  $Na_2CO_3$  (four parts) is added. When solution is complete, the liquid is concentrated to a small volume. Well-formed white, rhombohedral crystals appear, together with many-faceted, somewhat yellowish prisms.

PROPERTIES:

The rhombohedral crystals effloresce easily in air, becoming pure white and opaque. Water content: 34 moles of  $H_2O/mole$ . Readily soluble in  $H_2O$ .

The asymmetric, lustrous yellowish prisms do not effloresce in air. Water content: 26 moles of  $H_2O$ /mole. Soluble in  $H_2O$ .

The Free Acid  $I_2O_7 \cdot 12M_0O_3 \cdot aq$ .

The free acid can be prepared by ion exchange (see p. 1701). Thus, 25 g. of  $5 \operatorname{Na_2O} \cdot I_2 \operatorname{O_7} \cdot 12 \operatorname{MoO_3} \cdot \operatorname{aq}$ . is dissolved in 100 ml. of H<sub>2</sub>O. After passage through the column, the eluate is clear and colorless. Gas evolves during concentration in a vacuum desiccator. A bright-yellow crystalline compound is obtained. Yield: 22 g.

PROPERTIES:

Bright-yellow crystals; moderately soluble in cold H<sub>2</sub>O, readily soluble in hot.

## **REFERENCES:**

C. W. Blomstrand. Z. anorg. Chem. <u>1</u>, 10 (1892); G. Jander and D. Ertel. Unpublished experiments.

# 48-Vanadic Acid-2-Phosphates and

# 24-Vanadic Acid-2-Phosphates

It was concluded from chemical and physicochemical studies that the anions of these heteropolyacid compounds consist of phosphate ions and ions of octavanadic acid  $H_{10}(V_8O_{25} \cdot aq.)$  in varying molecular ratios [G. Jander and K. F. Jahr, Kolloid-Beihefte <u>41</u>, 324 (1935)]. Other authors obtained compounds which they classify as salts of a 12-vanadic-1-phosphoric acid  $H_7[PV_{12}O_{36} \cdot aq.]$  [A. Rosenheim and M. Pieck, Z. anorg. allg. Chem. <u>98</u>, 223 (1916); P. Souchay and S. Dubois, Ann. Chimie (12) <u>3</u>, 88 (1948)].

In addition to the compounds described here, there exists an immense number of salts of other compositions. The composition of the crystalline salt depends essentially on the molar ratio of phosphoric and vanadic acids in the starting solution, and also on the  $\rm H^+$  concentration, the nature of the cation, and the absolute concentration.

The brownish-red heteropolyacid compounds of this class, rich in vanadic acid, are designated in the older literature as "purpureophosphovanadates."

The Sodium Salt  $10 \operatorname{Na}_2 0 \cdot \operatorname{P}_2 0_5 \cdot 24 \operatorname{V}_2 0_5 \cdot \operatorname{aq}$ .

Prepared by combining solutions of NaVO<sub>3</sub>, Na<sub>2</sub>HPO<sub>4</sub> and HNO<sub>3</sub>. Thus, the solution used is 0.75M in sodium metavanadate (NaVO<sub>3</sub> · aq.), 0.315M in Na<sub>2</sub>HPO<sub>4</sub>, and 1.125M in HNO<sub>3</sub> (these quantities do not take the reaction into account); it therefore contains 2.38 moles of vanadic acid per mole of phosphoric acid. This deep-red solution is treated with 1/5 its volume of acetone and allowed to stand in the cold. After a while, the desired salt crystallizes out.

PROPERTIES:

Formula weight 5128.5. Small, dark-red octahedral crystals. Soluble in  $H_2O$ . Like all "purpureophosphovanadates," it is quite susceptible to hydrolysis; excessive heating of the solution must be avoided.

REFERENCE:

G. Jander and K. F. Jahr. Kolloid-Beihefte 41, 332 (1935).

The Barium Salt  $10 BaO \cdot P_2O_5 \cdot 24 V_2O_5 \cdot aq$ .

$$\begin{array}{rl} 10 \operatorname{Na_2O} \cdot \operatorname{P_2O_5} \cdot 24 \operatorname{V_2O_5} (\mathrm{aq.}) &+ & 10 \operatorname{Ba}(\mathrm{NO_3})_2 \\ & & 5128.5 & 2613.8 \\ & = & 10 \operatorname{BaO} \cdot \operatorname{P_2O_5} \cdot 24 \operatorname{V_2O_5} (\mathrm{aq.}) &+ & 20 \operatorname{NaNO_3} \\ & & 6041.2 & 1700.2 \end{array}$$

A solution prepared in the same way as described for the sodium salt (50 ml.) is treated with 100 ml. of  $0.375N \operatorname{Ba(NO_3)_2}$  solution and about 1/5 its volume of acetone. After standing for some time, the barium salt crystallizes out.

PROPERTIES:

Deep-red cubic crystals; poor solubility in H<sub>2</sub>O.

REFERENCE:

G. Jander and K. F. Jahr. Kolloid-Beihefte 41, 332 (1935).

The Potassium Salt  $11K_2O \cdot 2P_2O_5 \cdot 24V_2O_5 \cdot aq$ .

The Ammonium Salt  $5(NH_4)_2 O \cdot 2P_2 O_5 \cdot 24V_2 O_5 \cdot aq$ .

These are obtained from solutions of the sodium salt 10 Na<sub>2</sub>O  $\cdot$  P<sub>2</sub>O<sub>5</sub>  $\cdot$  24 V<sub>2</sub>O<sub>5</sub>  $\cdot$  aq. by means of KNO<sub>3</sub> or NH<sub>4</sub>NO<sub>3</sub>.

The same solution as described above in the case of the sodium salt (50 ml.) is treated with 100 ml. of  $0.375N \text{ KNO}_3$  solution or  $0.375N \text{ NH}_4\text{NO}_3$  solution and 1/5 its volume of acetone. After standing for some time, the desired salt crystallizes out.

PROPERTIES:

Formula weight of 11 K<sub>2</sub>O  $\cdot$  2 P<sub>2</sub>O<sub>5</sub>  $\cdot$  24 V<sub>2</sub>O<sub>5</sub> : 5685.5; formula weight of 5 (NH<sub>4</sub>)<sub>2</sub>O  $\cdot$  2 P<sub>2</sub>O<sub>5</sub>  $\cdot$  24 V<sub>2</sub>O<sub>5</sub> : 4909.9. The potassium salt crystallizes in deep red rhombohedra, the ammonium salt in six-sided columns. Both are soluble in H<sub>2</sub>O.

REFERENCE:

G. Jander and K. F. Jahr. Kolloid-Beihefte 41, 332 (1935).

1740

# SECTION 4

# Carbonyl and Nitrosyl Compounds F. SEEL

#### **General Introduction**

The classical method for the preparation of Mo, Fe, Co and Ni carbonyls consists of a direct reaction of CO with the respective metal at high pressures (150-200 atm.) and temperatures (100-200°C). Under these conditions steel is generally attacked by CO with formation of  $Fe(CO)_5$ , so that pressure vessels fully lined with a CO-resistant material (e.g., copper-silver alloy) must be used. Since such special autoclaves are not normally found in general-purpose chemical laboratories, this compilation of carbonyl syntheses will be restricted to preparative methods which are compatible with the usual laboratory apparatus, i.e., atmospheric-pressure syntheses or those requiring simple steel autoclaves.

With this goal in mind, several completely new methods were worked out { i.e., for Ni(CO)<sub>4</sub>, [Co(CO)<sub>4</sub>]<sub>2</sub>Hg, Co(CO)<sub>3</sub>NO}, while in the case of others [Fe(CO)<sub>4</sub>H<sub>2</sub>, Co(CO)<sub>4</sub>H] the apparatus used was improved. All methods were rechecked. For this we thank especially W. Hieber and his co-workers.

Descriptions of specific autoclaves for preparation of carbonyl compounds may be found in L. Mond, Z. anorg. Chem. <u>68</u>, 207 (1910) and W. Hieber, H. Schulten and R. Martin, Z. anorg. allg. Chem. 240, 261 (1939).

#### Chromium, Molybdenum, Tungsten Carbonyls

#### $Cr(CO)_6$ , $Mo(CO)_6$ , $W(CO)_6$

The hexacarbonyls of the chromium group are formed via reaction of CO with a suspension of anhydrous halides of Cr, Mo or W in a Grignard solution, followed by hydrolysis. The reaction mechanism has not yet been elucidated.

The reactor vessel f in Fig. 343 is a one-liter flask fitted with a two-hole rubber stopper. The dropping funnel t has a considerably enlarged tip to prevent plugging during the reaction. It is used for the addition of the Grignard solution (via a), as well as that of CO (at b). Stopcock h is a gas vent which remains normally closed during the reaction but which is occasionally opened to allow flushing the reactor with CO. Flask f is fitted exactly into the ice bath e, and the whole apparatus is vigorously shaken on a machine. To monitor the CO consumption, a standardized gasometer is connected to b via a drying train (whose last tube is filled with  $P_2O_5$ ).



Fig. 343. Preparation of hexacarbonyls of the chromium group. *a* and *b* inlet tubes; *e* ice bath; *f* reactor vessel; *t* dropping funnel. Note that the tip of funnel *t* should be enlarged.

The reactor flask f is filled with nitrogen. The metal chloride (10 g. of fine anhydrous CrCl<sub>3</sub> powder; 17 g. of sublimed MoCl<sub>5</sub>; or 20 g. of WCl<sub>6</sub> [0.05 moles]) is introduced, and the vessel is evacuated and filled with CO. A mixture of 50 ml. of anhydrous ether and 50 ml. of anhydrous benzene is added through the dropping funnel and the apparatus is then connected to the CO line. The Grignard reagent is prepared from 12 g. (0.5 moles) of Mg, 54 g. of  $C_{2}H_{5}Br$  and approximately 300 ml. of anhydrous ether. This solution is added to the metal chloride suspension first in portions of about 5 ml. each, later dropwise. The initiation of the CO reaction as well as its progress may be observed via a wash bottle containing some conc. H<sub>2</sub>SO<sub>4</sub> provided the stopcock of t is closed. The absorption of CO, which for reasons unknown occasionally slows down and then accelerates, is continued for about 4-6 hours after the addition of all of the Grignard reagent. The reaction absorbs on the average 7 liters and occasionally up to 9 liters of CO.

The reddish-brown reaction product is hydrolyzed by cautious addition to a mixture of ice and dilute  $H_2SO_4$ , and the mixture is then steam-distilled without prior removal of ether and benzene. The steam distillation is continued for 3-4 hours or as long as white needles of the carbonyl product are observed in the (descending) condenser. The organic layer (benzene-ether) in the distillate is separated and the aqueous phase extracted 3-4 times with fresh ether. The combined ether extracts are concentrated by distillation, keeping the temperature below 60°C, and the residue is allowed to crystallize in a refrigerator.

The yields of crude carbonyls are quite variable: in the case of  $Cr(CO)_6$  they are 2 g. maximum, while up to 3-4 g. of  $W(CO)_6$  may be isolated. Higher yields of  $Cr(CO)_6$  (up to 67%) are obtained in an autoclave under high CO pressure (35-70 atm.). To remove strongly adhering, odorous organic impurities, an immediate vacuum sublimation of the hexacarbonyls is recommended.

PROPERTIES:

Formula weight of  $Cr(CO)_6$ : 220.1; of  $Mo(CO)_6$ : 264.0; of  $W(CO)_6$ : 352.0. Colorless, strongly refractive orthorhombic crystals which are isomorphic among themselves; well soluble in inert organic solvents and sublimable.  $Cr(CO)_6$  melts at 149-50°C in a sealed tube. The hexacarbonyls are remarkably stable in comparison to all other metal carbonyls. Their vapors decompose above 120°C in a combustion tube, depositing the metals as mirrors.

**REFERENCES:** 

W. Hieber and E. Romberg. Z. anorg. allg. Chem. <u>221</u>, 321 (1935);
B. B. Owen, J. English, Jr., H. C. Cassidy and A. V. Dundon. J. Amer. Chem. Soc. <u>69</u>, 1723 (1947).

## Iron Pentacarbonyl

Fe(CO)<sub>5</sub>

Iron carbonyl is an industrial product which is prepared by classical carbonyl synthesis from CO and finely divided iron:

 $\begin{array}{rcl} {\rm Fe} &+& 5\ {\rm CO} &=& {\rm Fe}({\rm CO})_5 \\ 55.9 & 112.0\ l. & 195.9 \end{array}$ 

This material serves as the starting substance for work in the field of iron carbonyls; laboratory preparation is not worthwhile since it requires the special autoclaves mentioned on page 1741.

There is, however, a method for the preparation of small quantities of  $Fe(CO)_5$  in which a regular laboratory autoclave may be used; it starts with  $Fe(CO)_4I_2$  (see p. 1751):

$$\begin{array}{ccc} 5 \, {\rm Fe(CO)_4 I_2} + 10 \, {\rm Cu} = 10 \, {\rm CuI} + 4 \, {\rm Fe(CO)_5} + {\rm Fe} \\ 2108.5 & 636 & 783.6 \end{array}$$

An intimate mixture of equal weights of  $Fe(CO)_4I_2$  and copper dust is heated in a CO stream. The initial temperature is 40°C. It is then increased to 55°C and the  $Fe(CO)_5$  (30% yield) is condensed in a U tube at -50°C.

If an autoclave at about 10 atm. pressure is used, the yield becomes nearly quantitative since the decomposition reaction

$$Fe(CO)_{4}I_{9} = FeI_{9} + 4 CO$$

which occurs at atmospheric pressure, becomes impossible. Iron pentacarbonyl can then be distilled directly from the reaction vessel.

#### PROPERTIES:

At room temperature yellow, oily liquid,  $d^{20}$  1.46. Vapor pressure equation in the range of 0 to 102.7°C.: log p = 7.349--1681/T. At -25°C, solidifies to monoclinic needles; distills without decomposition at 102.6°C and 760 mm.; crit. temp. 285-288°C. Produces a metallic iron mirror on passage through a hot glass tube (200-350°C). Not altered in the dark; decomposed in light to Fe<sub>2</sub>(CO)<sub>9</sub> and CO (must be stored in dark bottles). Pyrophoric in air (caution!); burns to Fe<sub>2</sub>O<sub>3</sub>. Nearly insoluble in water; readily soluble in many organic solvents, especially benzene, petroleum ether, ether, glacial acetic acid and acetone.

#### REFERENCE:

W. Hieber and H. Lagally. Z. anorg. allg. Chem. 245, 295 (1940).

## Diiron Nonacarbonyl

## Fe<sub>2</sub>(CO)<sub>9</sub>

Formed during decomposition of Fe(CO)<sub>5</sub> by light:

 $2 \operatorname{Fe}(\operatorname{CO})_{5} = \operatorname{Fe}_{2}(\operatorname{CO})_{9} + \operatorname{CO}_{391.8} \quad 363.8 \quad 22.4 \ l.$ 

A solution of 20 g. of  $Fe(CO)_5$  in 40 ml. of glacial acetic acid (or acetic anhydride) is prepared and exposed to direct sunlight in an atmosphere of hydrogen or under vacuum. Very soon, turbidity and crystallization of  $Fe_2(CO)_9$  are observed. The nascent CO is removed by flushing with  $H_2$  or by reevacuation of the vessel. After several hours of illumination, the crystals are collected by filtration and washed with ethanol and ether. Minimum yield: 30% (5 g.). Further illumination of the mother liquors yields more  $Fe_2(CO)_9$ .

**PROPERTIES:** 

Shiny, orange hexagonal platelets;  $d^{18}$  2.085. Nearly insoluble in ether, petroleum ether and benzene; somewhat soluble in methanol, ethanol, and acetone; more readily soluble in pyridine. Stable at room temperature in dry air; on heating to 100-120°C, decomposes according to:

 $2 \operatorname{Fe}_{2}(\operatorname{CO})_{9} = 2 \operatorname{Fe}(\operatorname{CO})_{5} + 2 \operatorname{Fe} + 8 \operatorname{CO}$ 

SYNONYM:

Diiron enneacarbonyl.

REFERENCE:

E. Speyer and H. Wolf. Ber. dtsch. chem. Ges. 60, 1424 (1927).

# Triiron Dodecacarbonyl

 $[Fe(CO)_4]_3$  or  $Fe_3(CO)_{12}$ 

Oxidation of  $Fe(CO)_4H_2$  gives  $Fe(CO)_4$ . The former may be prepared and oxidized in one consecutive process.

A) PREPARATION OF Fe(CO)<sub>4</sub> H<sub>2</sub> SOLUTION:

 $\begin{array}{rrrr} {\rm Fe}({\rm CO})_5 \ + \ 2 \ {\rm NaOH} \ = \ {\rm Fe}({\rm CO})_4 {\rm H}_2 \ + \ {\rm Na}_2 {\rm CO}_3 \\ 195.9 \ & 80.0 \ & 169.9 \ & 106.0 \end{array}$ 

A two-liter flask is filled with  $N_{2^{\circ}}$  Then, 14 ml. (= 20 g.) of Fe(CO)<sub>5</sub> and 60 ml. of methanol are added and agitated until a complete mixture of both liquids has been obtained. At this point, 30 ml. of 50% sodium hydroxide is added and, to avoid decomposition of the product, the mixture is cooled to 0°C. The reaction of the base with the pentacarbonyl occurs at room temperature and is completed after several minutes of shaking. Immediately after the addition of the base, the solution shows a milky turbidity due

to the formation of sodium methyl carbonate, which is only slightly soluble. Exposure to air (it is not necessary to exclude the latter completely to obtain the product) soon leads to a deep reddishbrown color.

# B) OXIDATION WITH MnO<sub>2</sub>

An aqueous suspension of  $MnO_2$ , prepared from 35 g. of  $MnSO_4$ . 7  $H_2O$ , 5.6 ml. of  $Br_2$  and 20 g. of NaOH and purified by decantation, is added slowly with constant shaking to a freshly prepared solution of  $Fe(CO)_4H_2$ . The color of the reaction mixture changes immediately to a deep green. The excess  $MnO_2$  is dissolved by addition of  $FeSO_4$  in sulfuric acid or of NaHSO<sub>3</sub>. At the end, 100 ml. of sulfuric acid (1:1) is added, resulting in a vigorous evolution of gas. After completion of the gas evolution, the solution is refluxed on a water bath for about 30 minutes. This coagulates the  $Fe(CO)_4$ ; because it is hydrophobic, it floats on top of the solution as a dark green, crystalline mass. The product is collected on a fritted-glass filter, washed with hot, dilute  $H_2SO_4$ ,  $H_2O$ , ethanol, and petroleum ether, and weighed in a desiccator. Yield: 90% (15-16 g.).

PROPERTIES:

Deep-green, nearly black, strongly dichroic square plates. Insoluble in water; slightly soluble (dark green color) in organic solvents such as benzene, petroleum ether, ether and acetone; more soluble in  $Fe(CO)_5$  and  $Ni(CO)_4$ ; in contrast to the other iron carbonyls, very sensitive to air.  $d^{18}$  1.996; crystal structure: tetragonal.

**REFERENCES:** 

 W. Hieber. Z. anorg. allg. Chem. <u>204</u>, 171 (1932); R. B. King and F. G. A. Stone in: Inorg. Syntheses, Vol. VII, New York-London, 1963, p. 193.

**Cobalt Carbonyls** 

$$[Co(CO)_{4}]_{2}, [Co(CO)_{3}]_{4}$$

Decomposition of the hydride  $Co(CO)_4H$  above its melting point gives  $[Co(CO)_4]_2$ :

 $2 \operatorname{Co}(\operatorname{CO})_{4} \operatorname{H} = [\operatorname{Co}(\operatorname{CO})_{4}]_{2} + \operatorname{H}_{2}_{344.0}_{342.0} + \operatorname{H}_{22.4}_{1.}_{1.}$ 

The hydride  $Co(CO)_4H$  (see p. 1753) is slowly evaporated under CO from a bath at an initial temperature of -30°C, which may be left unattended. Beautiful crystals of orange-red  $[Co(CO)_4]_2$  remain as a residue.

Alternate method: Direct high-pressure synthesis from Co metal and CO [L. Mond, H. Hirtz and M. D. Cowap, Z. anorg. Chem. <u>68</u>, 215 (1910)].

PROPERTIES:

Orange crystals, m.p. 51°C. Insoluble in  $H_2O$ ; soluble in ethanol, ether and other organic solvents. Decomposes in air to the violet, basic Co carbonate. Melting of Co(CO)<sub>4</sub> results in a very slow decomposition to cobalt tricarbonyl [Co(CO)<sub>3</sub>]<sub>4</sub>; this reaction is clearly observable at 53°C and proceeds so fast at 60°C that it is complete after two days. After recrystallization from benzene, the tricarbonyl, formed according to the equation Co(CO)<sub>4</sub> = Co(CO)<sub>3</sub> + CO, consists of deep black crystals.

**REFERENCE**:

W. Hieber, F. Mühlbauer and E. A. Ehmann. Ber. dtsch. chem. Ges. <u>65</u>, 1090 (1932).

# Nickel Carbonyl

#### Ni(CO)<sub>4</sub>

Forms even under such mild conditions as the reaction of CO with an alkaline solution of NiS. However, the reaction of CO with a solution of nickel sulfoxylate-ammonia complex,  $NiSO_2 \cdot (NH_3)_x$ , is especially recommended. This solution is easily prepared from  $NiSO_4$ ,  $NH_3$  and  $Na_2S_2O_4$ .

$$\begin{array}{rl} \text{NiSO}_4 &+ & \text{Na}_2\text{S}_2\text{O}_4 &+ & (x+2) \text{ NH}_3 &+ \text{ H}_2\text{O} \\ (\cdot 7 \text{ H}_2\text{O}) & (\cdot 2 \text{ H}_2\text{O}) \\ \underline{280.9} & (\cdot 2 \text{ H}_2\text{O}) \\ &= & \text{NiSO}_2 \cdot (\text{NH}_3)_x &+ & \text{Na}_2\text{SO}_4 &+ & (\text{NH}_4)_2\text{SO}_3 \\ \\ \text{NiSO}_2 \cdot (\text{NH}_3)_x &+ & \text{H}_2\text{O} &+ & 4 \text{ CO} \\ &= & \text{Ni}(\text{CO})_4 &+ & (\text{NH}_4)_2\text{SO}_3 &+ & (x-2) \text{ NH}_3 \\ \\ \underline{89.6 l.} & 170.7 \end{array}$$

The apparatus shown in Fig. 345 comprises a large gas-liquid mixing flask to which the shaking vessel shown in Fig. 344 can be attached. This mixing flask is attached to a large U tube filled with pea-sized calcium chloride granules and  $P_2O_5$  deposited on glass beads. This tube is in turn attached to three condensing traps, two tees with stopcocks, and a bubble counter at the end of the train. The individual parts of the apparatus may be connected

by short pieces of rubber, provided glass touches glass; however, a ground-glass connection is required at s (this allows attaching the two condensing traps k, which are fused to each other, to a vacuum line).

The mixing flask is charged with a solution of 14.0 g. (0.05 moles) of NiSO<sub>4</sub>  $\cdot$  7 H<sub>2</sub>O in 400 ml, of water and 60 ml, of 25% aqueous ammonia. The carefully predried shaking vessel is charged with 12.5 g. of 80% Na<sub>2</sub>S<sub>2</sub>O<sub>4</sub>; then 30 ml. of concentrated ammonia solution and 80 ml. of water are added under flowing nitrogen. The materials are dissolved by shaking, and the shaking vessel is attached to the mixing vessel by means of a rubber stopper, as shown in the figure. At this point CO, carefully prepurified to remove traces of iron carbonyl and oxygen, is passed through the clear blue nickel (II) salt solution; after a few minutes, a dropwise flow of the ammonial dithionite solution is started. (Pressures must be constantly equalized.) The dithionite solution is added over a period of about 20 minutes. The Ni(CO)<sub>4</sub> forms instantly and is condensed in the traps by cooling the latter with Dry Ice-acetone (or ethanol) mixture; the unreacted CO may be burned at  $t_2$  after a reasonable flow rate has been established.



Fig. 344. Shaking vessel for work in the absence of air.

The nickel salt solution becomes nearly colorless in about 5 hours; a slight amount of decomposition, shown by the deposition of a mirrorlike layer of  $NiS_x$  on the glass walls, can not be avoided. Redistillation of the carbonyl (which condenses primarily in the first trap) in a stream of CO gives a completely pure product. Yield: 7.5 g. (5.5-6.0 ml.) of Ni(CO)<sub>4</sub> = 85-90%.

PROPERTIES:

Colorless liquid. M.p. -25 °C, b.p. 43 °C; d<sup>20</sup> 1.310. Crit. temp. approximately 200 °C, crit. pressure 30 atm. Insoluble in water, soluble in organic solvents. Readily oxidized in air;



Fig. 345. Apparatus for the preparation of nickel carbonyl, carbonyl hydrides, and nitroyl carbonyls. *b* shaking vessel from Fig. 344; *k* condensing traps

burns with a luminous flame when ignited. A mixture with air is explosive. A bright Ni mirror is formed on passage through a heated tube at 180-200 °C. The vapor is extraordinarily toxic.

REFERENCE:

W. Hieber, E. O. Fischer and E. Böckly. Z. anorg. allg. Chem. <u>269</u>, 308 (1952).

Dipyridine Chromium Tetracarbonyl, Tripyridine Chromium Tricarbonyl

Cr(CO)<sub>4</sub>py<sub>2</sub>, Cr(CO)<sub>3</sub>py<sub>3</sub>

Up to one half of the CO contained in metal carbonyls can frequently be replaced by pyridine, o,o'-dipyridyl and phenan-throline.

A mixture of 1 g. of  $Cr(CO)_{6}$  and 5 ml. of pure anhydrous pyridine is heated in a sealed tube for 2 hours at 200°C. A large fraction of the  $Cr(CO)_{6}$  separates unchanged on cooling the

deep-brown solution. The solution is filtered into a small distilling apparatus and most of the pyridine is quickly distilled off in a stream of  $N_2$ . On cooling the residue, crystals of  $Cr(CO)_4py_2$  separate out; addition of petroleum ether increases the yield.

If 10-15 ml. of pyridine is used in the above experiment and the heating is repeated after discharge of the CO, the product is  $Cr(CO)_3py_3$ . Pyridine can be substituted into  $Mo(CO)_6$  more readily than into  $Cr(CO)_6$ ; the  $Mo(CO)_6$  is converted into  $Mo(CO)_3py_3$  on heating with pyridine for several hours at 135 °C.

PROPERTIES:

 $Cr(CO)_{4}py_{2}$ : Blunt, yellowish-brown prisms. On heating, loses pyridine and turns brownish-black.  $Cr(CO)_{3}py_{3}$ : Yellowish-red to red prisms, stable in air; gives up pyridine very readily.

REFERENCE:

W. Hieber and F. Mühlbauer. Z. anorg. allg. Chem. <u>221</u>, 341 (1935).

# o-Phenanthroline Nickel Dicarbonyl

#### $Ni(CO)_2C_{12}H_8N_2$

$$\begin{array}{rcl} \mathrm{Ni}(\mathrm{CO})_4 &+ & \mathrm{C_{12}H_8N_2} &= & \mathrm{C_{12}H_8N_2} \cdot \mathrm{Ni}(\mathrm{CO})_2 &+ & 2 \operatorname{CO} \\ 170.7 & & 180.2 & & 294.9 & & 44.8 \, l. \end{array}$$

Readily formed from solutions of equimolar quantities of the components in acetone or absolute ethanol. The almost immediate CO generation is preceded by a blood-red color of the reaction mixture. As the gas evolution increases, the compound separates in long needles (of the order of 1 cm.). After suction-filtration, the compound is washed with absolute ethanol, followed by petro-leum ether.

PROPERTIES:

Ruby-red needles with high luster; very stable. Decomposed by air only in the presence of moisture (slow process, accompanied by a green color).

REFERENCE:

W. Hieber, F. Mühlbauer and E. A. Ehmann. Ber. dtsch. chem. Ges. <u>65</u>, 1098 (1932).

1750

#### Iron Tetracarbonyl Halides

#### Fe(CO)<sub>4</sub>X<sub>2</sub>

I. The simplest method consists of reacting the halogens with  $Fe(CO)_5$  in organic solvents.

Pure Fe(CO)<sub>4</sub>Cl<sub>2</sub> can be obtained only at -20 °C by slow passage of Cl<sub>2</sub> through a 1M solution of Fe(CO)<sub>5</sub> in petroleum ether; the slight decomposition of the Fe(CO)<sub>5</sub> does not affect the purity of the product. The product (yield ~ 60%) is a yellow powder, which loses CO at room temperature and turns gray.

 $Fe(CO)_4Br_2$  is prepared by slowly adding the reactants to a 2M petroleum ether solution, while cooling in ice and conducting the reaction under anhydrous conditions. The product is a reddishbrown powder. Yield: 75%.

 $Fe(CO)_4I_2$  is obtained in anhydrous ether solution using a slight excess of  $Fe(CO)_5$  [the ether solution is 2M in  $Fe(CO)_5$  and 0.5M in  $I_2$ ]. On evaporation, the compound separates in large, black crystals. Yield: quantitative.

PROPERTIES:

The iron carbonyl halides are light-sensitive: water converts them to the corresponding Fe (II) salt solutions (the chloride and bromide react instantaneously, the iodide only upon heating). The thermal decomposition of iron carbonyl halides is a convenient way to produce fine powders of anhydrous Fe(II) halides.

II.  $Fe(CO)_{4I_2}$  can be obtained from anhydrous  $FeI_2$  and CO at pressures above 6.3 atm. Thus, it can be prepared in ordinary laboratory autoclaves.

$$FeI_2 + 4 CO = Fe(CO)_4I_2$$
  
309.7 89.6 l. 421.7

The preparation proceeds in anhydrous etheral solution, using 50 ml. of ether/g. of FeI<sub>2</sub>; the air is displaced from the autoclave by evacuating and flushing with CO (which may be taken directly from a cylinder). While the yield is quantitative, the duration of the CO absorption depends on the surface/volume ratio in the solution. Thus, in unagitated systems the reaction may occasionally take several days.

**REFERENCES:** 

- W. Hieber and G. Bader. Ber. dtsch. chem. Ges. <u>61</u>, 1717 (1928); W. Hieber and H. Lagally. Z. anorg. allg. Chem. <u>245</u>, 295 (1940); W. Hieber and A. Wirsching. Ibid. <u>245</u>, 35 (1940).
- II. W. Hieber and H. Lagally. Ibid. 245, 305 (1940).

# Iron Tetracarbonyl Dihydride

#### Fe(CO)<sub>4</sub>H<sub>2</sub>

Solutions of alkali metal, alkaline earth metal, and ammonium salts of  $Fe(CO)_4H_2$  may be produced by shaking solutions or suspensions of the corresponding hydroxides with  $Fe(CO)_5$  in the absence of air. Addition of acid liberates the hydride.

 $\begin{array}{rll} {\rm Fe}({\rm CO})_5 &+ 3\;{\rm NaOH} &= \; [{\rm Fe}({\rm CO})_4 {\rm H}]{\rm Na} \;+\; {\rm Na}_2 {\rm CO}_3 \;+\; {\rm H}_2 {\rm O} \\ 195.9 & 120.0 & 191.9 \\ & \; [{\rm Fe}({\rm CO})_4 {\rm H}]{\rm Na} \;+\; {\rm H}_2 {\rm SO}_4 \;=\; {\rm Fe}({\rm CO})_4 {\rm H}_2 \;+\; {\rm NaHSO}_4 \\ & \; 191.9 & 98.1 & 169.9 \end{array}$ 

The reaction may be conducted in the apparatus (Figs. 344 and 345) described for the preparation of Ni(CO)<sub>4</sub>. After complete evacuation with an oil pump, the 200-ml. shaking vessel is charged with 7 ml. of Fe(CO)<sub>5</sub> (10 g., 0.1 moles), then with 25 g. of NaOH in 50 ml. of boiled water; the vessel walls are rinsed with some water and it is filled with oxygen-free nitrogen or CO. The reaction is complete after 5 hours of intensive mixing on a shaking machine. If no oxygen was present, a completely homogeneous, light-yellow solution of the sodium salt of Fe(CO)<sub>4</sub>H<sub>2</sub> is produced. At this point, the mixing vessel is charged with 50 ml. of 50% sulfuric acid, the shaking vessel is attached, and the entire apparatus is flushed with oxygen-free CO (which can be ignited at  $t_2$  after a reasonable flow rate has been attained).

The mixing vessel is then cooled to -10 to -15 °C (ice-salt mixture) and the cold traps to -80 °C (Dry Ice-acetone). The hydride solution from the shaking vessel is then decomposed by slow, dropwise addition to the mixing vessel. The liberated hydride is carried by the CO stream into the cold traps, where it condenses as a solid (if this fails to occur, the temperatures of the cooling baths are too high or the CO flow too high). Upon termination of the decomposition the apparatus is flushed with CO until a "carbonyl flame" is no longer visible on burning at  $t_2$ ; this takes about one hour [see the section on Ni(CO)<sub>4</sub>]. The hydride collected in the first two condensation traps is not yet

pure. It contains the products of its own decomposition,  $Fe(CO)_5$ and  $Fe(CO)_3$ , and occasionally its oxidation product  $Fe(CO)_4$ . Repeated fractionation in high vacuum at -40 °C, with condensation at -80 °C, is necessary for complete purification. The removal of the last traces of  $Fe(CO)_5$ , which is frequently still present in small quantities (since it is re-formed during the distillation due to the extreme instability of the hydride), is achieved by repeated rapid distillation into a liquid-nitrogen-cooled vessel.

Since the isolation and purification of the free carbonyl hydride require considerable time, the acid decomposition should be started at the beginning of a full working day.  $Fe(CO)_4H_2$  can be stored indefinitely, provided it is placed in suitable containers which are evacuated and cooled with liquid nitrogen.

PROPERTIES:

At the temperature of liquid nitrogen, a completely colorless, crystalline substance. M.p. -70°C. The autodecomposition

$$2 \operatorname{Fe(CO)}_{4} \operatorname{H}_{9} = \operatorname{Fe(CO)}_{5} + \operatorname{Fe(CO)}_{8} + 2 \operatorname{H}_{9}$$

of the water-white, mobile liquid is already observed at  $-10^{\circ}$ , as is indicated by a slight reddish color [Fe(CO)<sub>3</sub>], but is completed only at higher temperature. The presence of even traces of Fe(CO)<sub>3</sub> and Fe(CO)<sub>5</sub> in the hydride is indicated by a weak red or yellow color; absence of color (only below  $-10^{\circ}$ C) is a sensitive test for the purity of the product. The gaseous hydride has an extremely nauseating odor.

**REFERENCES:** 

W. Hieber and H. Vetter. Z. anorg. allg. Chem. 212, 145 (1933).

An ether-soluble iron carbonyl hydride  $Fe_4(CO)_{13}H_2$  was isolated from the reaction of pyridine-iron carbonyl with acids [W. Hieber and R. Werner, Chem. Ber. <u>90</u>, 286 (1957)].

# Cobalt Tetracarbonyl Hydride

# Co(CO)<sub>4</sub>H

An alkaline suspension of  $Co(OH)_2$  absorbs CO in the presence of a small amount of cyanide to form a solution of the alkali salt
of  $Co(CO)_4H$ . The hydride is then liberated from this solution with acid.

$$\begin{array}{rcl} 2 \operatorname{Co}(\mathrm{NO}_3)_2 &+& 11 \operatorname{CO} &+& 12 \operatorname{KOH} \\ \stackrel{(\cdot \ 6 \ H_2 \mathrm{O})}{582.1} && 246.4 \ l. && 673.2 \\ && = & 2 \operatorname{Co}(\mathrm{CO})_4 \mathrm{K} \ +& 4 \operatorname{KNO}_3 \ +& 3 \operatorname{K}_2 \mathrm{CO}_3 \ +& 6 \operatorname{H}_2 \mathrm{O} \\ && 420.2 \\ && & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & \\ && & & & \\ && & & \\ && & & \\ && & & & \\ && & & & \\ && & & \\ && & & & \\ && & & \\ && & & & \\ && & & & \\ && &$$

The same apparatus as is shown in Figs. 344 and 345 (see preparation of  $Ni(CO)_4$  is used. Since the acid decomposition of the alkaline hydride solution produces considerable foaming, a second mixing vessel, placed in series with the first one, is The evacuated shaking vessel is charged with recommended. 14.6 g. (0.05 moles) of  $Co(NO_3)_2 \cdot 6 H_2O$  in 30 ml. of water, 22.4 g. of KOH in 22 ml. of water, 10 ml. of water, 3.2 g. of KCN in 6 ml. of water (in this order), and finally with 10 ml. of water. Then stopcock  $h_1$  is connected to a gas-liquid mixing vessel equipped with a fritted disk and containing an alkaline pyrogallol The vessel is connected to a calibrated gasometer solution. containing 6 liters of CO. After flushing the inlet piece via stopcock  $h_2$ , the gas is admitted to the shaking vessel; the latter is flushed briefly via  $h_3$ ,  $h_3$  is closed again, and the assembly is shaken on a machine under COpressure. The CO absorption may be observed via the pyrogallol wash bottle: it varies with the intensity of shaking and even quantitative absorption may occasionally be achieved. During the absorption, the initially brown suspension is converted into a yellow solution which, in the case of absorption of the theoretical amount of CO, is nearly free of suspended matter and consists of a solution of Co(CO)<sub>4</sub>K.

The free hydride is isolated in the same way as described in the preceding preparation of  $Fe(CO)_4H_2$ . On the same scale of preparation, the acid decomposition requires 50 ml. of 50% sulfuric acid. The hydride is finally sublimed from a bath maintained at  $-30^{\circ}C$  to a cooler condensing trap.

#### PROPERTIES:

The pure solid hydride is completely colorless and crystalline; it melts at -26.2 °C to a light-yellow solution which changes to dark yellow at slightly higher temperature {beginning of decomposition to  $[Co(CO)_4]_2$ }. In gas form  $Co(CO)_4$ H has a nauseating odor and is extremely toxic. REFERENCES:

W. Hieber and H. Schulten. Z. anorg. allg. Chem. <u>232</u>, 29 (1937);
 A. A. Blanchard and P. Gilmont. J. Amer. Chem. Soc. <u>62</u>, 1192 (1940).

#### Iron Carbonyl Mercury

#### Fe(CO)<sub>4</sub>Hg

 $Fe(CO)_5 + HgSO_4 + H_2O = Fe(CO)_4Hg + CO_2 + H_2SO_4$ 195.9 296.7 368.5

A solution of 7.5 g. (0.025 moles) of  $HgSO_4$  in 50 ml. of 10%  $H_2SO_4$  is shaken with 3.5 ml. (5 g.) of  $Fe(CO)_5$  at room temperature to yield a dark-yellow, microcrystalline precipitate of  $Fe(CO)_4$  Hg. The precipitate is collected by filtration, washed several times with dilute  $H_2SO_4$ , 2-3 times with 2N HCl, then with water and acetone, and dried in vacuum.

PROPERTIES:

Insoluble in common solvents. Decomposes at about  $150^{\circ}$ C into Fe, Hg and CO. Reaction of HgCl<sub>2</sub> with Fe(CO)<sub>5</sub> gives Fe(CO)<sub>4</sub> Hg<sub>2</sub>Cl<sub>2</sub>.

**REFERENCE:** 

H. Hock and H. Stuhlmann. Ber. dtsch. chem. Ges. <u>61</u>, 2097 (1928); <u>62</u>, 431 (1929).

#### Cobalt Carbonyl Mercury

#### [Co(CO)<sub>4</sub>]<sub>2</sub>Hg

Produced by reaction of solutions of a salt of  $Co(CO)_4H$  with Hg (II) salts. The optimum preparative method requires work in an ammonia solution.

 $2 [Co(CO)_4]NH_4 + HgCl_2 = [Co(CO)_4]_2Hg + 2 NH_4Cl$ 378.0 271.5 542.6

The potassium salt of  $Co(CO)_4H$  is prepared as described on page 1754. However, only one half the amount is needed. The shaking vessel is evacuated, and 25 ml. of a saturated  $NH_4Cl$  solution (corresponding to about 10 g. of  $NH_4Cl$ ) and 3.5 g. of  $HgCl_2$  in 100 ml. of water are added in that order. The instantly formed ochre precipitate consists of  $[Co(CO)_4]_2Hg$ ,  $HgNH_2Cl$  and traces of Hg. The reaction flask is immediately opened; the precipitate is collected by filtration (without any special protective measures) and washed twice with water, dilute hydrochloric acid (to dissolve  $HgNH_2Cl$ ) and finally again with water. The crude product is dried in a desiccator and dissolved in some acetone; water is added until persistent turbidity. The product is left in a refrigerator to crystallize.

Alternate method: Precipitation from a solution of  $[Co(CO)_4]NH_4$ , as prepared by the dithionite method [see section on Ni(CO)<sub>4</sub>, p. 1747 f.] [W. Hieber, E. O. Fischer and E. Böcky, Z. anorg. alleg. Chem. <u>269</u>, 308 (1952)].

**PROPERTIES:** 

Orange needles; very stable; insoluble in water and dilute nonoxidizing acids; readily soluble in ethanol, ether, acetone, benzene and other similar solvents [in contrast to  $Fe(CO)_4Hg$ ]. An easily prepared compound and an excellent starting material for the preparation of other cobalt carbonyl compounds. Thus,  $[Co(CO)_4]_2Hg$  may be converted by alkali sulfides into alkali salts of  $Co(CO)_4H$ , from which the hydride itself and  $Co(CO)_4$  may be obtained.

In a similar manner, other heavy metal derivatives of  $Fe(CO)_4H$ and  $Co(CO)_4H$  may be obtained by this double decomposition. Derivatives of  $Co(CO)_4H$  and Zn, Cd, Hg, In, Tl are obtained directly from the metals and CO in high-pressure syntheses.

REFERENCES:

W. Hieber and H. Schulten. Z. anorg. alig. Chem. <u>232</u>, 24 (1937);
 W. Hieber and E. Fack. Ibid. <u>236</u>, 101 (1938); W. Hieber and
 U. Teller. Ibid. <u>249</u>, 43 (1942).

## Ethylenediamine Iron Carbonyl

## [Fe en<sub>3</sub>] [Fe<sub>2</sub>(CO)<sub>8</sub>]

This and the next compound are complex salts of polynuclear iron carbonyl hydrides.

 $3 \operatorname{Fe}(\operatorname{CO})_5 + 3 \operatorname{en} = [\operatorname{Fe} \operatorname{en}_3] [\operatorname{Fe}_2(\operatorname{CO})_8] + 7 \operatorname{CO}_{587.7} 180.2 571.8 156.8 l.$ 

This compound can be successfully prepared only under completely anhydrous conditions. The ethylenediamine (not ethylenediamine hydrate) and the solvent pyridine must be completely free of water. The presence of pyridine is essential to the reaction. The apparatus shown in Fig. 346, consisting of the reaction vessel a, fritted-glass filter attachment g, and dropping funnel and adapter t, is recommended.



Fig. 346. Preparation of ethylenediamine iron carbonyl. *a* reaction vessel; *g* fritted-glass filter; *h* stopcock; *s* ground joint adapter; *t* dropping funnel attachment.

With vessel *a* pointed downward, a solution of 1.4 g. of the diamine in 20 ml. of pyridine is mixed with 8 g. of  $Fe(CO)_5$ ; vessel *a* is closed with a ground-glass stopper at *s* and connected to the atmosphere via the stopcock *h* and a wash bottle filled with conc.  $H_2SO_4$ . The reaction mixture is then heated on a water bath to 80°C for four to five hours. After about one hour of heating the (now red) solution starts evolving gas bubbles and continues to do so until the reaction is finished. At end of the reaction, the solution is cooled, the dropping funnel *t* is connected to *s*, and the product, which forms in copious quantities, is filtered through *g*. It is washed on *g* with pyridine and anhydrous ether; for a final purification, it is again triturated with ether and refiltered. Yield: 2.6 g. (60% on the basis of the diamine).

SYNONYM:

Triethylenediamine iron (II) octacarbonyl diferrate (II).

PROPERTIES:

Brick-red monoclinic (or triclinic) and very shiny prisms; quite stable; insoluble in organic solvents, including pyridine.

REFERENCES:

W. Hieber and F. Sonnekalb. Ber. dtsch. chem. Ges. <u>61</u>, 563 (1928);
 W. Hieber, J. Sedlmeier and R. Werner. Chem. Ber. <u>90</u>, 278 (1957).

## Pyridine Iron Carbonyl

## [Fe py<sub>6</sub>] [Fe<sub>4</sub>(CO)<sub>13</sub>]

 $5 \operatorname{Fe}(\operatorname{CO})_{4} + 6 \operatorname{py} = [\operatorname{Fe} \operatorname{py}_{6}] [\operatorname{Fe}_{4}(\operatorname{CO})_{13}] + 7 \operatorname{CO}_{839.4} 474.6 1117.9 156.8 \, l.$ 

This substance must be prepared in complete absence of air and under nitrogen; this is best done in the apparatus shown in Fig. 346.

The bulb is charged with 5 g. of  $Fe(CO)_4$ ; the latter is then freed of the always present traces of  $Fe_3O_4$ . This is done by pouring over it some methanol and then heating with 20% HCl on a water bath for four to six hours (no oxygen may be present). The solution is suction-filtered through the fritted glass and washed with dilute HCl, dry methanol and dry ether; the residue is dried in a high vacuum. For this final drying, the dropping funnel is replaced by a stopper at s and the apparatus connected at h to a cooled trap and the vacuum pump.

The purified  $Fe(CO)_4$  is then reacted at 0°C with 6 ml. of dry and air-free pyridine, the reaction taking place in the evacuated apparatus. The reaction is accompanied by foaming and evolution of CO, and ends in a short while, the green color of the  $Fe(CO)_4$ solution turning intense red.

After 0.5 hour the product is collected by filtration, washed briefly with pyridine, then with dry petroleum ether and absolute ether, and dried in a high vacuum. Yield: 2.3 g. (70%), the remainder being retained in the mother liquor.

Alternate method: Direct synthesis from  $Fe(CO)_5$  and pyridine in a sealed tube at 120-140 °C.

SYNONYM:

Hexapyridine iron (II) tridecacarbonyl tetraferrate (II).

PROPERTIES:

Nearly black or deep-red crystals, extremely pyrophoric.

**REFERENCES**:

W. Hieber and E. Becker. Ber. dtsch. chem. Ges. <u>63</u>, 1414 (1930);
 W. Hieber and F. Mühlbauer. Ibid. <u>65</u>, 1088 (1932); W. Hieber and R. Werner. Chem. Ber. <u>90</u>, 286 (1957).

#### Potassium Nitrosyl Tricarbonyl Ferrate

#### [Fe(CO)<sub>3</sub>NO]K

 $Fe(CO)_5 + KNO_2 = [Fe(CO)_3NO]K + CO + CO_2$ 195.9 85.1 209.0

The reactor in this case is a one-liter, three-neck flask fitted with stirrer, reflux condenser and gas inlet tube. To start with, the air is completely displaced with very pure nitrogen; then, 45 g. of KNO<sub>2</sub> is dissolved in 400 ml. of methanol with vigorous stirring (the  $KNO_2$  is premelted and then cooled on a cold porcelain surface to give small droplets). Following this, 67 ml. (0.5 moles) of freshly distilled  $Fe(CO)_5$  is added via the condenser and the mixture is cautiously heated to 30-35 °C. The reaction starts in a short while and is accompanied by vigorous evolution of gas; since the reaction is highly exothermic, it must occasionally be moderated by cooling in cold water. The gas evolution slows down in about three hours; the mixture is heated at 60°C for 30 minutes and then cooled. Finally, the stirrer and the reflux condenser are replaced by stoppers, and the solvent and unreacted carbonyl compound are distilled off in aspirator vacuum. To shorten this distillation somewhat, it is permissible to heat slightly (but very carefully).

The dry crude product can be used immediately for the preparation of  $Fe(NO_2)(CO)_2$ . However, there may arise a need to purify it, especially to remove excess nitrite and decomposition products. In this case, the crude product is extracted (in the absence of air and light) in a Soxhlet apparatus with 200 ml. of ether until the reflux is colorless. Evaporation of the extract under reduced pressure yields a bright-orange mass. Addition of toluene or xylene to the ether solution gives the salt as fine crystals. Yield: 60 g. of crude product, which on careful workup yields 45 g. of pure substance.

#### **PROPERTIES:**

Orange-yellow salt, very sensitive to light and air, especially when in solution. All operations, including the purification, should be conducted in a darkroom under red light. The product is kept in a vacuum desiccator lined with black paper; the desiccator is preflushed several times with nitrogen, and is then evacuated. When stored under such conditions, the preparation is stable for some time.

The sodium salt is obtained in nearly quantitative yield by reaction of  $Fe(CO)_5$  with  $NaNO_2$  in absolute methanol in the presence of 2 molar equivalents of  $NaOCH_3$  per mole of  $Fe(CO)_5$ .

REFERENCES:

W. Hieber and H. Beutner. Z. Naturforschg. <u>15b</u>, 323 (1960);
 M. J. Hogsed. C.A. <u>53</u>, 9592 (1959), U. S. Pat. 2,865,707 (1957).

## Iron Dinitrosyl Dicarbonyl

#### $Fe(NO)_2(CO)_2$

This compound is prepared in the same apparatus as nickel carbonyl (p. 1747 ff.; Figs. 344 and 345). First, the shaking vessel, which in this case is used as a dropping funnel, is charged with 42 g. (0.2 moles) of  $[Fe(CO)_3NO]K$ ; after evacuation, a solution of 17 g. of  $KNO_2$  or 14 g. of  $NaNO_2$  in 150 ml. of water is added. Then, the shaking bulb is attached to the gas-liquid mixing vessel and the entire apparatus flushed with a moderately fast stream of air-free  $CO_2$ .

The brown nitrosyl carbonyl vapor appears the moment the solution is allowed to flow into the mixing vessel; the vapor is condensed in the Dry Ice-cooled traps, where it deposits as a bright orange coating. To speed up the transfer of the vapor to the traps, the reaction flask is heated in a water bath. However, the flask temperature should not exceed  $35^{\circ}$ C. If, as may happen (especially at the start of the experiment), some of the vapor condenses in the connecting tubes, it is driven into the cold traps by gentle heating with a hair dryer. The brown vapor disappears after a while, the preparation is stopped, and the CO<sub>2</sub> in the apparatus is displaced with pure nitrogen. The produce may be resublimed onto  $P_{2O_5}$  in high vacuum; from there, it may be driven into ampoules, which are then stored in a freezer and protected from light to avoid decomposition. The yield is approximately 20-25 g. (60-70%).

PROPERTIES:

Beautiful deep-red crystals. M.p.  $18.5^{\circ}$ C. The liquid has a tendency to supercool; decomposes at  $50^{\circ}$ C. Insoluble in water, soluble in organic solvents; readily oxidized by air. Can be distilled without extensive decomposition only at temperatures below  $15^{\circ}$ C.

REFERENCES:

W. Hieber and J. St. Anderson. Z. anorg. allg. Chem. <u>208</u>, 238 (1932); 221, 132 (1933); F. Seel. Ibid. 269, 40 (1952).

# 1761

#### Cobalt Nitrosyl Tricarbonyl

#### Co(NO) (CO)<sub>3</sub>

Prepared in the same way as  $Fe(NO)_2(CO)_2$ , that is, by reaction of the solution of  $Co(CO)_4K$  obtained in the cyanide process with nitrite and  $CO_2$ .

 $\frac{[Co(CO)_4]K + KNO_2 + 2CO_2 + 2H_2O = CO(NO)(CO)_3 + 2KHCO_3}{210.1 85.1 171.0}$ 

**PROPERTIES**:

Cherry-red, very volatile liquid; m.p. -1.05 °C, b.p. 48.6 °C, decomposition temperature 55 °C. Insoluble in water and very stable if kept under water; miscible in all proportions with ethanol, ether, acetone, benzene and other organic solvents.

**REFERENCES**:

W. Hieber and J. St. Anderson. Z. anorg. allg. Chem. <u>208</u>, 238 (1932); <u>221</u>, 132 (1933); F. Seel. Ibid. <u>269</u>, 40 (1952).

**Dinitrosyl Cobalt Halides** 

 $(NO)_2CoCl$ ,  $(NO)_2CoBr$ ,  $(NO)_2CoI$ 

The apparatus consists of a 70-cm.-long, 2-cm.-O.D. glass tube surrounded by a 15-cm.-long metal block with a thermometer well. The block is heated with a gas burner, the temperature being controlled automatically by means of a relay-actuated valve in the gas line, which in turn is tripped by a bimetallic element in the thermowell. The reactor tube is connected to a tee so that either dry N<sub>2</sub> or NO<sub>2</sub> may be passed through. The NO is generated from NaNO<sub>2</sub> and 20% H<sub>2</sub>SO<sub>4</sub>, washed free of NO<sub>2</sub> with 50% KOH, and dried in a train consisting of flask filled with conc. H<sub>2</sub>SO<sub>4</sub>, CaCl<sub>2</sub>, NaOH and P<sub>2</sub>O<sub>5</sub> (in that order). All parts of the apparatus are either connected with ground joints or fused together.

The raw material for the iodine compound consists of about 2 g. of anhydrous CoI<sub>2</sub> placed in a porcelain boat, which is then inserted into the front part of the glass tube; the latter is then closed off with a wash bottle containing conc. H<sub>2</sub>SO<sub>4</sub>. The traces of moisture carried in with the CoI<sub>2</sub> are then removed by heating to 130°C (by means of the metal block) while passing through a stream of N<sub>2</sub>. After cooling, the N<sub>2</sub> is displaced with NO and the temperature is again raised to 70-80°C. At this point, the reaction of NO with CoI<sub>2</sub> is so rapid that the pressure in the apparatus drops to below atmospheric; at the same time copious amounts of iodine are given off. The substance sinters and takes on a violet sheen. The temperature is now raised to 105°C and maintained at that level for 15-20 hours, that is, until the iodine vapors are displaced by brown vapors of the nascent (NO)<sub>2</sub>CoI. The final residue in the porcelain boat is a viscous, blackishbrown mass.

The product must be sublimed. Thus, the boat is transferred, in a countercurrent stream of  $N_2$ , to the other end of the reactor tube and the temperature is raised to 115 °C. The beginning of the sublimation is noticeable by a brown deposit; at a later stage, beautiful, flexible, deep brown-black, glittering crystals, up to 15 mm. long, are formed.

The chlorine and bromine compounds are prepared in the same way, except that a halogen-trapping metal (such as Zn dust or Co powder) must be added in 20% excess.

PROPERTIES:

Formula weight of (NO)<sub>2</sub>CoCl: 182.4; (NO)<sub>2</sub>Br: 226.9; (NO)<sub>2</sub>CoI: 273.9. After sublimation, these compounds form beautiful blackbrown needles with a diamondlike glitter; these are often 1 cm. long. Melting points: (NO)<sub>2</sub>CoCl: 101°C; (NO)<sub>2</sub>CoBr: 116°C; (NO)<sub>2</sub>CoI: 131°C. The freshly prepared compounds are stable for some time in air; however, the crystals lose their surface sheen in several hours and then decompose over a period of days with loss of NO. The solubility in water increases in the sequence I-Br-Cl (partial decomposition).

In an analogous fashion iron (II) halides yields the dinitrosyl iron halides, (NO)<sub>2</sub>FeX, upon which Roussin's salts are based.

REFERENCES:

W. Hieber and R. Marin. Z. anorg. allg. Chem. <u>240</u>, 241 (1939);
 W. Hieber and R. Nast. Ibid. <u>244</u>, 23 (1940).

#### Sodium Dinitrosyl Thioferrate\*

#### $Na[(NO)_2FeS] \cdot 4 H_2O$

$$\begin{array}{rcl} & 2\,(\mathrm{NH}_4)[(\mathrm{NO})_7\mathrm{Fe}_4\mathrm{S}_3] \ + \ 6\,\mathrm{NaOH} \\ & (\cdot\,\mathrm{H}_2\mathrm{O}) \\ & 113.1 & 24.0 \\ & = \ 6\,\mathrm{Na}[(\mathrm{NO})_2\mathrm{Fe}\mathrm{S}] \ + \ 2\,\mathrm{Fe}(\mathrm{OH})_3 \ + \ \mathrm{N}_2\mathrm{O} \ + \ 2\,\mathrm{NH}_3 \ + \ \mathrm{H}_2\mathrm{O} \\ & & (\cdot\,4\,\mathrm{H}_2\mathrm{O}) \\ & 145.8 \end{array}$$

A mixture of 10 ml. of 10% NaOH with 3 g. of Roussin's black ammonium salt is prepared and heated on the water bath at  $80^{\circ}$ C; the heating is continued until the ammonia odor disappears (in about 15 minutes). The Fe(OH)<sub>3</sub> precipitate is removed by suction filtration through a fritted-glass funnel and the reddishbrown solution is evaporated over CaCl<sub>2</sub> under reduced pressure. It is left standing for one day; beautiful black-red crystals separate. These are collected on a fritted-glass funnel, washed with 0.1% sodium hydroxide solution, and dried between filter papers.

SYNONYM:

Roussin's red sodium salt.

PROPERTIES:

Formula weight of Na[(NO)<sub>2</sub>FeS] 170.92; of Na[(NO)<sub>2</sub>FeS]  $\cdot$  4 H<sub>2</sub>O 242.99. Reddish-black crystals or reddish-brown powder, soluble in water and ethanol; insoluble in ether. Upon removal of excess base, the compound is converted in a short time to the black Roussin's salt.

**REFERENCES:** 

O. Pawel. Ber. dtsch. chem. Ges. <u>12</u>, 1953 (1879); <u>15</u>, 2607 (1882).

<sup>\*</sup>The nomenclature used here and in the following preparations is justified by the nature of these complex salts, in which the NO group is bound in the same way as in the nitrosyl carbonyls. See F. Seel, Z. anorg. allg. Chem. 249, 308 (1942).

## Ammonium Heptanitrosyl Trithiotetraferrate

## $NH_4[(NO)_7Fe_4S_3] \cdot H_2O$

A solution of 8 g. of NaNO<sub>2</sub>\* in 40 ml. of water is mixed with a solution of (NH<sub>4</sub>)<sub>2</sub>S, prepared by saturating 5 ml. of 22% ammonium hydroxide with H<sub>2</sub>S adding 5 ml. of ammonium hydroxide of the same concentration, and siluting with 30 ml. of water. The final mixture is heated to the boil and thus becomes a darkbrown. Now, a solution of 20 g. of FeSO<sub>4</sub> • 7 H<sub>2</sub>O in 160 ml. of water is added at once and the mixture is quickly reheated to a vigorous boil. The reaction starts even before the boiling point is reached [precipitation of  $Fe(OH)_3$  and S is evident from the color change to black and brown]. Almost simultaneously the mixture starts evolving nitrous fumes. If a good yield is desired, it is essential that the gas evolution be suppressed by addition of 25 ml. of ammonium hydroxide (in small portions) during the entire boiling operation. After boiling for 15 minutes, the hot solution is filtered as quickly as possible through two Buchner funnels (moderate vacuum). The small crystals of Roussin's black ammonium slat already begin to precipitate in the filtrate during the filtration. To obtain larger crystals, the filtrate is placed in a hot water bath, heated until the salt is completely dissolved, and allowed to cool in the bath. Yield: about 1.7 g.

SYNONYM:

Roussin's black ammonium salt.

PROPERTIES:

Formula weight of  $NH_4[(NO)_7 Fe_4S_3] \cdot H_2O$ : 565.7. Hard, monoclinic crystals with a diamondlike glitter; soluble in water, giving dark-brown solutions; stable to 80°C.

The corresponding alkali salts are obtained by a similar procedure; however no excess of nitrite is necessary in this case.

<sup>\*</sup>This is about double the theoretical amount; this compensates for the decomposition of the  $NH_4NO_2$  present in the solution.

REFERENCES:

O. Pawel. Ber. dtsch. chem. Ges. <u>12</u>, 1953 (1879); <u>15</u>, 2607 (1882).

# Ethyl Dinitrosyl Thioferrate

#### $[(NO)_2 Fe SC_2 H_5]_2$

The shaking vessel shown in Fig. 344 is charged with 27.8 g. (0.1 moles) of FeSO4 • 7 H<sub>2</sub>O and evacuated. Then, 140 ml. of boiled water is aspirated in, the salt is dissolved by shaking, and finally a solution of 11.2 g. of KOH and 3.1 g. of C<sub>2H5SH</sub> (3.7 ml., 0.05 moles) in 25 ml. of water is added. The apparatus is now connected to a calibrated gas-measuring flask containing 2.5 liters of NO stored over some solid KOH; the gasometer is then shaken on a machine. The initially sirupy, off-color slurry of  $Fe(OH)_2$  and  $Fe(SC_2H_5)_2$  becomes a deep olive-green liquid. Toward the very end of the NO absorption (which goes very fast at the beginning and takes about 1.5 hours), the color changes to a light brown. A small amount of deep-black crystals are evident at the bottom of the flask and sometimes at the surface of the liquid. The NO is displaced with  $N_2$  and the flask is opened. The black crystals can be separated from the Fe(OH)<sub>3</sub>, which has a lower specific gravity, by decantation and slurrying; however, better yields (up to 80%) are obtained by centrifugation, washing in the centrifuge tube (once with absolute ethanol and 3-4 times with ether), and recrystallization of the substance from hot absolute ethanol with slow cooling.

#### **PROPERTIES:**

Glittering black monoclinic crystals, m.p. 78°C. Insoluble in water; soluble with difficulty in ethanol; more readily soluble in ether; readily soluble in  $CS_2$ ,  $CHCl_3$  and  $C_6H_6$ , giving yellowish-red solutions.

REFERENCES:

K. A. Hofmann and O. F. Wiede. Z. anorg. Chem. 9, 300 (1895);
 H. Reihlen and A. von Friedolsheim. Liebigs Ann. 457, 71 (1927).

## Potassium Dinitrosyl Thiosulfatoferrate

#### $K[(NO)_2FeS_2O_3] \cdot H_2O$

 $\begin{array}{l} 2 \operatorname{FeSO}_4 + 4 \operatorname{K}_2 S_2 O_3 + 4 \operatorname{NO} = 2 \operatorname{K}[(\operatorname{NO})_2 \operatorname{FeS}_2 O_3] + \operatorname{K}_2 S_4 O_6 + 2 \operatorname{K}_2 \operatorname{SO}_4 \\ (\cdot 7 \operatorname{H}_2 O) \\ 556.0 \\ 761.2 \\ 89.6 \ l. \\ 516.2 \end{array}$ 

This derivative of Roussin's red salt is prepared by shaking a mixture of the concentrated aqueous solutions of 28 g. (0.1 moles) of  $FeSO_4 \cdot 7 H_2O$  and 40 g. of  $K_2S_2O_3$  under NO, as described in the previous preparation. During the first hour, the gas is absorbed especially rapidly and the solution turns intensely brown. Later,  $K[(NO)_2FeS_2O_3] \cdot H_2O$  separates out in leaflets of brass-like glitter. The substance is collected by filtration, washed with ethanol, and dried in vacuum over conc. sulfuric acid.

PROPERTIES:

Formula weight of  $K[(NO)_2FeS_2O_3] \cdot H_2O$ : 258.1. Only slightly soluble in cold and warm water; decomposes in boiling water.

The same procedure is used to prepare the sodium salt,  $Na[(NO)_2FeS_2O_3]$ . In this case, crystallization is slower due to the greater solubility of the product.

Finally, the same method may be used for the preparation of the corresponding cobalt and nickel compounds  $K_3[(NO)_2Co(S_2O_3)_2]$ , and  $K_3[(NO)Ni(S_2O_3)_2]$ .

REFERENCES:

K. A. Hofmann and O. F. Wiede. Z. anorg. Chem. 8, 319 (1895);
 W. Manchot. Ber. dtsch. chem. Ges. 59, 2445 (1926).

### Potassium Nitrosyl Cyanomolybdate

 $K_4[(NO)Mo(CN)_5] \cdot H_2O$ 

$$\begin{array}{rrrr} MoO_3 &+ 5 \, \text{KCN} &+ 6 \, (\text{NH}_3\text{OH})\text{Cl} &+ 5 \, \text{KOH} \\ 144.0 & 325.6 & 417.0 & 280.5 \\ & & = \, \text{K}_4[(\text{NO})\text{Mo}(\text{CN})_5] \,+\, 2 \, \text{N}_2 \,+\, \text{NH}_3 \,+\, 6 \, \text{KCl} \,+\, 12 \, \text{H}_2\text{O} \\ & & \quad (\cdot \, \text{H}_2\text{O}) \\ & & \quad 412.4 \end{array}$$

The reaction is based on reduction with  $NH_2OH$  and disproportionation of the latter into NOH and  $NH_3$ . Thus, 5 g, of

 $MoO_3$  powder is treated with 10 ml. of a solution of 25 g. of KOH in 20 ml. of water; the mixture is stirred to the point where everything just dissolves. A saturated aqueous solution containing 20 g. of KCN is then added and the mixture filtered through a mediumporosity fritted-glass funnel. Then, 17.5 g, of (NH<sub>3</sub>OH)Cl is added to the filtrate and the mixture stirred until the (NH<sub>3</sub>OH)Cl is dissolved. The red solution is now heated on a water bath for 30 minutes; then, an additional 10 ml. of conc. KOH is added. At this point, the red color changes to a light yellow and then slowly turns to violet. The appearance of the violet color is accompanied by the evolution of NH<sub>3</sub> and is quickly followed by the separation of the violet NO compound. The latter is collected (after cooling) by filtration, washed with alcohol and ether, redissolved in a minimum of hot water, and quickly filtered into cold, 50% potassium hydroxide. The deep-violet, crystalline compound reprecipitates (the mixture must sometimes be left standing overnight); it is washed with ethanol and ether and dried in vacuum. Yield: 40%.

**PROPERTIES:** 

Hygroscopic; decomposes in air, becoming lemon-yellow. May be stored indefinitely under nitrogen; in vacuum, may be heated up to 180°C without decomposition or loss of water of crystallization.

Readily soluble only in water; insoluble in all the usual organic solvents such as ethanol, ether, benzene, acetone, pyridine, chloroform,  $CCl_4$  and  $CS_2$ . The aqueous solution is quite unstable and decomposes after a short time with loss of color.

REFERENCE:

W. Hieber, R. Nast and G. Gehring. Z. anorg. allg. Chem. 256, 173 (1948).

### Potassium Nitrosyl Cyanomanganate

## $K_{3}[(NO)Mn(CN)_{5}]$

$$\begin{array}{rcl} K_{3}[Mn(CN)_{6}] &+& 2\,(NH_{3}OH)Cl &+& 3\,KOH\\ 328.3 & & 139.0 & 168.3\\ &=& K_{3}[(NO)Mn(CN)_{5}] \,+& KCN \,+& 2\,KCl \,+& NH_{3} \,+& 4\,H_{2}O\\ && 332.3 \end{array}$$

A solution of 16.4 g. (0.05 moles) of  $K_3[Mn(CN)_6]$  in 100 ml. of a 15% KCN solution is treated (in this order) with 7 g. of solid  $(NH_3OH)Cl$  and a solution of 8.4 g. of KOH in 20 ml. of water. The mixture is then slowly heated, whereby it becomes brown, then violet. At this point flocculent  $Mn(OH)_3$  appears occasionally due to hydrolytic cleavage of the  $K_3[Mn(CN)_6]$ . The end of the reaction is indicated by a dark-violet color of the solution and termination of the NH<sub>3</sub> evolution. After cooling, the solution is made weakly acidic by addition of acetic acid, and 18.5 g. of  $Mn(CH_3CO_2)_2 \cdot 4 H_2O$  in 60 ml. of water is added to give a copious but readily filtered rose-red precipitate of  $Mn_3[(NO)Mn(CN)_5]_2$ :

$$\begin{array}{rcl} K_{3}[(NO)Mn(CN)_{5}] &+& 1^{1}/_{2} Mn(CH_{3}CO_{2})_{2} \\ &&& (\cdot 4 H_{2}O) \\ &&& 367.7 \\ && = & \frac{1}/_{2} Mn_{3}[(NO)Mn(CN)_{5}]_{2} &+& CH_{3}CO_{2}K \\ &&& 307.4 \end{array}$$

The precipitate is carefully washed and digested with a solution of 35 g. of  $K_2CO_3$  in 120 ml. of water:

 $\frac{1}{2} Mn_3[(NO)Mn(CN)_5]_2 + \frac{1}{2} K_2CO_3 = K_3[(NO)Mn(CN)_5] + \frac{1}{2} MnCO_3$ 307.4 307.3 332.3

After slight heating the solution may be filtered, if necessary; it is then made weakly acidic with acetic acid, and a large excess of ethanol is added. On standing for a few days, the initially flocculent precipitate changes to fine, deep-violet crystals, which are collected and washed with ethanol. When kept over  $P_2O_5$ , the substance loses all water of crystallization and is converted to the anhydrous  $K_3[(NO)Mn(CN)_5]$ . Yield: 80-90%.

REFERENCE:

W. Hieber, R. Nast and E. Proeschel. Z. anorg. allg. Chem. <u>256</u>, 167 (1948).

### Sodium Nitrosyl Cyanoferrate

 $Na_2[(NO)Fe(CN)_5] \cdot 2 H_2O$ 

 $\begin{array}{l} K_4 [Fe(CN)_6] + 6 \, HNO_3 = \, H_2 [(NO)Fe(CN)_5] + 4 \, KNO_3 + NH_4 NO_3 + CO_2 \\ ( \cdot 3 \, H_2 O) \\ 422.4 \end{array}$ 

 $\begin{array}{l} H_2[(NO)Fe(CN)_5] \,+\, Na_2CO_3 \,=\, Na_2[(NO)Fe(CN)_5] \,+\, H_2O \,+\, CO_2 \\ (\,\cdot\, 2\, H_2O) \\ 298.0 \end{array}$ 

A 400-ml. beaker is used to dissolve 40 g. of  $K_4[Fe(CN)_6]$ . 3  $H_2O$  in 60 ml. of water (slight heating). Then, 64 ml. of nitric acid (d 1.24) is added (stirring). The mixture is digested on a water bath at moderate temperature until a test drop of the brown solution reacts with  $FeSO_4$  solution to give a dark green (rather than blue) precipitate. After standing for 1-2 days, the mixture is just neutralized with  $Na_2CO_3$  (an excess must be avoided). The neutralized solution is heated to the boil, filtered and quickly concentrated to a small volume. After cooling, an equal volume of ethanol is added to precipitate most of the KNO<sub>3</sub>. This is separated by filtration, and the solution is quickly reconcentrated to remove the ethanol. The dark-red solution yields crystals on standing; these are suction-filtered and washed with some cold water. Further crystalline material is obtained by repeating the evaporation of the mother liquors.

SYNONYM:

Sodium nitroprusside.

PROPERTIES:

Ruby-red orthorhombic-bipyramidic crystals. One part is soluble in 2.5 parts of water at 16°C.

REFERENCES:

L. Vanino. Handb. d. präp. Chem. [Handbook of Preparative Chemistry], Inorganic Part, Stuttgart, 1925, p. 355; R. Wild. Arch. Pharm. <u>131</u>, 26 (1855).

## Sodium Carbonyl Cyanoferrate

### Na<sub>3</sub>[(CO)Fe(CN)<sub>5</sub>]

Easily prepared via reaction of CO with  $Na_3[Fe(OH_2)(CN)_5]$ , which in turn is obtained from sodium nitroprusside.

a) 
$$Na_{2}[(NO)Fe(CN)_{5}] + (NH_{3}OH)Cl + Na_{2}CO_{3}$$
  
 $(\cdot 2 H_{2}O)$   
 $298.0 69.5 106.0$   
 $= Na_{3}[Fe(OH_{2})(CN)_{5}] + NaCl + N_{2}O + CO_{2} + H_{2}O$   
 $272.9 58.5$ 

A solution of 20 g. of sodium nitroprusside and 10 g. of  $Na_2CO_3$ in 80 ml. of water is prepared; the reaction flask is placed in ice and 7 g. of (NH<sub>3</sub>OH)Cl is some water is added. Gas evolution begins immediately and the solution turns a greenish brown. After one hour, the product is precipitated as a brown tar with three volumes of ethanol. By repeated solution in water and reprecipitation with methanol,  $Na_3[Fe(OH_2)(CN)_5]$  is obtained as a yellow powder. It is important to keep the temperature below 5°C to avoid formation of  $Na_4[Fe(CN)_6]$  and iron hydroxides.

b)  $Na_{3}[Fe(OH_{2})(CN)_{5}] + CO = Na_{3}[(CO)Fe(CN)_{5}] + H_{2}O$ 272.9 22.4 l. 282.9

The freshly prepared aquo complex (13 g.) is dissolved in 35 ml. of boiled water and introduced into an evacuated three-neck flask fitted with a vacuum-tight mercury-seal stirrer, a dropping funnel, and gas inlet and outlet tubes. By repeated flushing and evacuation, the flask is filled with carbon monoxide (prewashed with an alkaline pyrogallol solution). When vigorously stirred, the solution starts to absorb CO at a fast rate; after 24 hours and absorption of 98% of the stoichiometric quantity of CO, the reaction is complete. The flask must be protected from light during the reaction. The product is precipitated from the greenish-blue solution by addition of 200 ml. of ethanol containing 0.5 g. of NaOH; it is filtered in air. After washing with some absolute ethanol, the complex is obtained in analytical purity.

Since Na<sub>3</sub>[(CO)Fe(CN)<sub>5</sub>] is somewhat soluble in ethanol, it is recommended to work up the aqueous-alcoholic filtrate. To this end the filtrate is evaporated to dryness at 12 mm. and 40-50 °C (bath temperature), the residue dissolved in the minimum amount of water, and all Fe(OH)<sub>3</sub> filtered out. The filtrate is concentrated to crystallization in a vacuum desiccator over conc. H<sub>2</sub>SO<sub>4</sub>; a very small quantity of mother liquor should remain. The crystals are collected by filtration, washed with some ethanol, and dried at 110 °C. Yield: up to 90%.

PROPERTIES:

Pale-yellow needles, surprising stable.

REFERENCE:

W. Hieber, R. Nast and C. Bartenstein. Z. anorg. allg. Chem. <u>272</u>, 32 (1953).

1770

## SECTION 5

# Alloys and Intermetallic Compounds

G. BRAUER

### General Remarks

The usual laboratory preparation of alloys consists of fusion of the metallic components. This method allows a simple control over the quantities of reagents so as to reach the desired composition. If the changes of the phase diagram of the metal system are known as a function of the temperature, this method also allows, in most cases, the preparation of definite intermetallic compounds. Occasionally, however, it is difficult to obtain the required homogeneity in the product because some of the reagents may burn, evaporate or react with the fusion vessel.

By comparison, other methods for the preparation of alloys are less used in the laboratory, although in special cases the optimum methods may involve reduction (chemical or electrolytic) of metallic compounds. In addition, some intermetallic compounds are best obtained as residues remaining after the corresponding basic alloys are dissolved. However, a knowledge of the temperature-induced transformations of the phase diagram is of the greatest importance in all cases; thus, the literature references below must be consulted, if at all possible.

Due to the enormous number of possible and known alloys and intermetallic compounds, a full description of all preparative methods is, of course, impossible. It is even less feasible to cite all the most important compounds separately. This section therefore contains only a selection of typical laboratory procedures; these are examples which may be adapted to other cases, even if the latter are unrelated. Only a few individual preparations are given in detail.

Further, it should be pointed out that preparative methods for metallic compounds overlap those for semimetals and nonmetallic compounds. For this reason, many of the methods cited here are derived from those for nonmetallic compounds; others, which stem from metallurgy, may also be applied to other substances. REFERENCES:

Collected phase diagrams of metallic systems:

M. Hansen and K. Anderko. Constitution of Binary Alloys, New York-Toronto-London, 1958; Landolt-Börnstein. Zahlenwerte und Funktionen [Numerical Values and Functions], Vol. II, Part 3, 6th Ed., Berlin-Göttingen-Heidelberg, 1956; W. Hume-Rothery et al. Metallurgical Equilibrium Diagrams, London, 1952; E. Janecke. Kurzgefasstes Handbuch aller Legierungen [Short Handbook of Alloys], Heidelberg, 1949; T. Lyman (Amer. Soc. Metals). Metals Handbook, Cleveland, 1948 (Binary and Ternary Alloys); J. L. Haughton. Bibliography of the Literature Relating to Constitutional Diagrams of Alloys, London, 1942; M. von Schwarz. Metall- und Legierungskunde [Metals and Alloys], Stuttgart, 1929 (Binary, Ternary and Quaternary Alloys).

Preparation of Alloys by the Use of Heat

## Purity of the Starting Materials

Except for special cases where some purification is achieved by the vaporization that occurs at relatively high temperatures, one should not expect that the product alloys will be purer than the starting metals. Therefore, the latter should be as pure as possible and should contain a minimum of dissolved impurities ("internal" impurities). The "external" impurities also cannot be neglected. Thus, oxide layers must be removed by scraping or grinding, or by chemical etching with suitable acids. Industrial metals comminuted by mechanical means (powders, shavings) are frequently contaminated by traces of lubricants. These must be removed by extraction with organic solvents; otherwise, they tend to interfere with the alloy formation and form carbides. Water and all organic solvents must be removed by careful drying.

The optimum methods sometimes involve metal hydrides rather than pure metals. The procedure is useful mainly in the case of metals that form stable hydrides (alkali and alkaline earth metals, Ti to Th, V to Ta, Pd). The hydrides are readily reduced to powders and the contact of the latter with the other components of the alloy is much better than it would be otherwise. The thermal decomposition of the hydrides proceeds so easily that the formation of alloys is not only not slower than in the case of pure metals, but is faster due to the small particle size of the material. In addition, the hydrogen liberated from the hydride may reduce the oxide impurities. One disadvantage inherent in the use of hydrides is that the commercial materials are usually less pure than the corresponding metals.

### Form of the Starting Materials

The starting material may consist of chunks, ingots, shavings or powders. Large chunks have relatively small surface areas, thus introducing fewer "external" impurities; they usually leave less material on the container walls. On the other hand, mixtures of large chunks may sometimes be difficult to reduce to a homogeneous melt, especially if the components of the alloy differ greatly in density or melting points. While homogeneous mixtures of powders already in the solid state can be prepared, the oxide skin frequently prevents junction of the particles even when sufficient heat is applied; in addition, powders have a greater tendency to cling to the container walls, again because of surface oxides. The metal hydrides may be found advantageous in this case, as mentioned above. The formation of alloy from powders, shavings or thin wires is greatly improved by pressing the mixture into pellets prior to heating (suitable dies are described in Part I, p. 103). (They are made from shape-retaining "oil-tempered" steel and are hardened only after machining and careful fitting of the die and the matrix.)

Metals which readily acquire a surface oxide layer may be cut into a potlike shape on a lathe. The other alloy components can then be hammered into the hollow to assure an intimate contact from the very start.

When two components with widely differing melting points are fused, the fusion pot should be arranged so that the lowermelting metal must run through the higher-melting one.

## Preparation of Starting Mixtures

It is desired to obtain the desired alloy composition by weighing out theoretical quantities of the components but, for various reasons, this composition cannot always be achieved that simply. Frequently a number of successive preliminary experiments must be carried out, whereby one gains the necessary experience.

The most important causes of deviation of the product from the desired composition are losses of metal by vaporization, oxidation or side reactions with the material of the fusion pot. In such cases the expected losses of a component are balanced by adding an extra quantity of that component to the starting mixture. A rough approximation of the extra quantity required is obtained from the fact that when the preparation is conducted properly and in closed crucibles (see below), not even the very reactive mixtures of alkali metals lose more than 5% of the starting weight. An especially clean way to measure out alkali metals consists in melting the latter in small, sealed glass ampoules, from which they can easily be removed by remelting (see the section on Alkali Metals, pp. 961-967).

## **Crucible and Ampoule Methods**

The alloy components, weighed out with the above considerations in mind, are combined by fusion in crucibles or ampoules. Some method must always be devised to minimize losses due to burning or vaporization. In simple cases, where open vessels are used, this is achieved by covering the charge with a protective layer of a salt or salt mixture which also melts in the process. Alternatively, the mixture may be protected by a blanket of an inert gas; crucibles may be closed by a lid and ampoules by fusing the constricted neck. If the closure is gas-tight, some inert or reducing gases may be included and a vacuum may even be maintained.

Several low-melting salts and salt mixtures suitable for laboratory use are given in Tables 1-3 (for further references, see Guertler [1]). Many such protective agents are commercially available for industrial use and they can also be employed.

#### Table 1

Melting	Points	of Some	Salts	Suitable	$\mathbf{for}$	Use	in	
	Protective Layers							

Salt	M.p., °C	Salt	М.р., °С
LiNO <sub>8</sub> NaNO <sub>8</sub> KNO <sub>9</sub> LiCl MgCl <sub>2</sub> Na <sub>2</sub> B <sub>4</sub> O; CaCl <sub>2</sub> KCl NaCl KF	255 307 334 613 708 741 772 776 801 880	LiF Na <sub>2</sub> SO4 KBO2 BaCl2 NaBO3 K <sub>2</sub> SiO3 NaF Na <sub>2</sub> SiO3 CaF2	870 884 947 962 966 976 988 1088 1360

Hygroscopic salt mixtures sometimes react with the molten alloys, evolving hydrogen and interfering in the reaction. This effect can be reduced by adding KOH.

The type of protective atmosphere depends on the metals of the alloy. Hydrogen is frequently used, except when large quantities of hydride-forming alkali, alkaline earth or rare earth

#### Table 2

Wt. %	Salt I	Wt. %	Salt II	м.р., °С	Wt. %	Salt I	Wt. %	Salt II	М.р., °С
73 55.3 46 57	KNO₃ NaNO₃ LiCl LiCl	27 44.7 54 43	LiNO <sub>3</sub> LiNO <sub>3</sub> KCl KCl	132 208 352 580	32.8 51 77.8 45	KCl LiBO <sub>2</sub> NaCl NaCl	67.2 49 22.2 55	BaCl <sub>2</sub> NaBO <sub>2</sub> BaCl <sub>2</sub> KCl	645 648 654 660
61.4 44 12 32.8	KCl NaCl LiF NaCl NaCl	38.6 56 88 67.2 66 3	MgCl <sub>2</sub> MgCl <sub>2</sub> LiCl CaCl <sub>2</sub>	426 430 485 500	35.8 72 51 50 32 4	LiF NaCl KCl Na <sub>2</sub> CO <sub>3</sub>	64,2 28 49 50 67.6	MgF <sub>2</sub> NaF K <sub>2</sub> SO <sub>4</sub> K <sub>2</sub> CO <sub>3</sub> KF	669 675 690 690
45.8 73.5 44.2 63 35 46.6 32.8	KF KCl Na₂SO₄ KCl NaCl NaCl NaCl	54.2 26.5 55.8 37 65 53.4 67 2	AlF <sub>3</sub> CaCl <sub>2</sub> Li <sub>2</sub> SO <sub>4</sub> KF Na <sub>2</sub> CO <sub>3</sub> Na <sub>4</sub> P <sub>2</sub> O <sub>7</sub> Na <sub>2</sub> SO <sub>4</sub>	565 600 601 605 620 620 623	32.4 35.4 63.7 52.8 61 21.5 90.4 87.8	LiF LiF NaF NaF NaF KF BaFa	64.6 36.3 47.2 39 78.5 9.6 12.2	AlF <sub>3</sub> AlF <sub>3</sub> CaF <sub>2</sub> MgF <sub>2</sub> AlF <sub>3</sub> MgF <sub>2</sub>	710 715 810 815 985 835 890
26 85	KCl CaCl <sub>9</sub>	74 15	$CaCl_2$ $CaF_2$	640 644	78.8	BaF <sub>2</sub>	21.2	MgF <sub>2</sub>	930

#### Melting Points of Some Binary Salt Mixtures with Uniform Melting Points

metals are present; in other cases, nitrogen is used, except when nitride-forming Li, Be, Mg, Ca, Sr and Ba, or the rare earth metals, Ti, Zr, Hf, Th, V, Nb and Ta are present. If no carbides can form, CO may be used to advantage; however,  $CO_2$  and  $SO_2$ may occasionally oxidize the metals at high temperatures. Noble gases, especially argon, which is commercially available in cylinders at 150-200 atmospheres pressure, are the best but also the most expensive protective agents. For real protection the gas should be very pure: oxygen is undesirable even in traces. Gas purification methods are given in various sections of this handbook (H<sub>2</sub>: p. 111 ff.; N<sub>2</sub>: p. 457 ff.; noble gases: p. 82 ff.). Occasionally, H<sub>2</sub>, N<sub>2</sub> and Ar are available in high purity (99.99%) from commercial sources, sometimes on special order.

High, narrow crucibles are preferred. Useful crucible materials are a) metals, b) ceramics and c) glasses (for ampoules).

A) METALS:

For obvious reasons, only high-melting metals which do not tend to form alloys are suitable for crucibles. Iron and various types of steel, as well as molybdenum and tontolum, are frequently used. Molybdenum is very serviceable but also much more expensive and less easily worked than Fe. These metals are preferentially used for smelting alloys of the "B" group of

#### G. BRAUER

elements and for the extremely reactive alkali and alkaline earth metals (see Table 4).

#### Table 3

## Melting Points of Some Ternary Salt Mixtures with Uniform Melting Points

Wt. %	Salt I Wt.		Salt II Wt.		Salt III	M.p., °C	
16.4 76.4 24 5 53.3 10.1 15.9 20.5	NaCl BaCl <sub>2</sub> NaCl NaCl AlF <sub>3</sub> AlF <sub>3</sub> AlF <sub>3</sub>	24.6 14.0 37 9 13.2 34.4 26.7 51 7	$\begin{array}{c} \text{KCl} \\ \text{KCl} \\ \text{KCl} \\ \text{KCl} \\ \text{Ca}F_2 \\ \text{Ca}F_2 \\ \text{Ca}F_2 \\ \text{Ca}F_2 \end{array}$	59.0 9.6 39 86 33.2 55.5 57.4 27.8	BaCl <sub>2</sub> Na <sub>2</sub> CO <sub>3</sub> Na <sub>2</sub> CO <sub>3</sub> Na <sub>2</sub> B <sub>4</sub> O <sub>7</sub> NaF NaF NaF NaF NaF	540 542 580 640 705 780 825 1095	

#### Table 4

### Metallurgical Classification of Elements

																н	He
Li											Be	В	С	N	0	F	Ne
Na											Mg	Al	Si	P	s	Cl	Ar
к	Ca	Sc	Ti	v	Cr	Mn	Fe	Co	Ni	Cu	Zn	Ga	Ge	As	Se	Br	Kr
Rb	Sr	Y	Zr	Nb	Mo	Tc	Ru	Rh	Pd	Ag	Cd	In	Sn	Sb	Te	I	x
Cs	Ba	R.E.	Hf	Та	w	Re	Os	Ir	Pt	Au	Hg	Tl	Pb	Bi	Po	At	Em (Rn)
Fr	Ra	Ac	Th	Pa	U												
Alk and kali ear met	kali- d al- line rth etals									ΒE	leme	nts					

Metallic Pt, Ag and Ni, which are normally chemically inert and thus are used as vessel materials under corrosive conditions, cannot be used in crucibles because of their intrinsic and pronounced tendency to form alloys.

The preparation of alloys of the so-called transition metals virtually always requires ceramic crucibles (see also below).

A gas-tight seal for an iron crucible can be obtained in a number of ways. An iron lid may, for instance, be put on and sealed in place by means of flanges; the lid may also be in the form of a threaded cap or plug, which yield a firmly closed circular seal. Two types of screw caps are shown in Fig. 273, p. 990. Tubular crucibles with fitting lids (or plugs) can frequently be welded shut for a gas-tight seal. In this case, the plug closely fits the L.D. of the crucible and has the shape of a tube with one end closed. Examples of this are given in Fig. 347. Even though the crucible may contain a protective gas blanket, the gas-filled space above the material to be alloyed should be kept at a minimum. With this in mind, the inner cylinder serving as a plug in the crucible of Fig. 352b should be hammered down (after charging) as far as possible, sawed off near the rim of the outer crucible and then welded to the outer rim. A plug shaped as a hollow cylinder is easier to weld at the rim than a solid plug. The hollow plug should be slightly tapered (into a cone) near its upper rim. Alternately, its rim should be turned down slightly after it is driven in (it thus forms a flange surface). In either case, the aim is to close off the seam and prevent welding gases from penetrating into the crucible. If generation of a large amount of heat is expected on welding and if it is undesirable to trigger the reaction until the crucible is tightly sealed, its lower section may be cooled in water during welding. This would have to be done, for instance, in welding crucibles containing the very volatile alkali and alkaline earth metals when they are to be alloved with "B" metals.



Fig. 347. Tubular steel crucible. When small quantities of alloy are needed (for instance, in crystal structure studies) the following dimensions have proved useful: O.D. 20-25 mm., wall thickness 1-2 mm., height 70-90 mm.

Prior to closure of the crucible, the protective gas may be introduced through a narrow tube. A better method is to put the



Fig. 348. A method for evacuating a metal crucible and filling it with a protective gas blanket.

charged crucible, with the lid loosely in place, into a large-diameter glass tube, which is closed at one end and fitted with a ground glass joint and stopcock (Fig. 348). The tube is then alternately evacuated and filled with the protective gas. This arrangement displaces the air very efficiently.

After fusion and cooling, the iron crucibles are opened by sawing off the top and bottom, and the alloy is then punched out of the open cylinder. In another method, the crucible is placed on a lathe, clamped at the stopper end, and its wall turned down to 0.1-0.3 mm. This thin wall can then be stripped off with a pair of pliers in the same manner as the top of a can of sardines. Since this can be done very rapidly, even those alloys which are very sensitive to air can be isolated without too much damage and can be rapidly transferred to a storage vessel filled with a protective gas. The method is also useful in cases where the alloy ingot adheres firmly to the crucible walls because of local

welding.

Alloys which are extremely sensitive to air, especially those with high concentrations of alkali and alkaline earth metals, require special methods for the removal of the ingot from the iron crucible [2]. In the device of Zintl and Harder (Fig. 349) a tubular iron crucible can be opened while completely surrounded by a protective gas.

The iron tube r is connected via the standard-taper joint s to a source of pure  $N_2$  as well as to other devices for further treatment of the alloy. The tubular crucible t, whose wall has already been machined down on a lathe (see above) so that it is now thin, is positioned in r by means of the screws a in such a way that its bottom may be sawed off through the slit c while still under  $N_2$  (the length of the circumferential slit c is equal to only 2/3 of the diameter of tube r). After the crucible bottom is sawed off, the crucible is pushed to the right by means of a thick wire (which passes through an axial hole in the cover plate e) and repositioned by means of screws a. The top of the crucible (that is, the plug) is now sawed off; the cover plate e is taken off for

a moment and the debris removed. The alloy is now contained in an iron tube (casing) which is open at both ends. The iron filings still adhering to the tube and ingot are tapped out. The alloy can now be pushed out of the casing by a steel rod passed through the hole in e. In certain cases the casing tube (including the alloy inside) may be used directly in further workup, e.g., an extraction. The protective gas should escape through only one hole at a time; the slit c for the saw can be closed off by collar d, the hole in eby means of a rubber stopper. This apparatus was originally used for the preparation of Na-Pb and Na-Sn alloys.



Fig. 349. Opening of a tubular iron crucible in the absence of air. r iron tube with standard-taper joint s; t crucible; a screws to fasten the crucible in place (nine such screws are arranged, in groups of three, along the circumference of r, the angles between the screw axes being 120°, as indicated in the insert); c slit for the introduction of a hacksaw; d collar for closing the slit when not in use; e cover plate for the tube r.

In the arrangement of Klemm and Dinkelacker, described in greater detail in Fig. 353 and on p. 1788 f., complete removal of the ingot from the tube is unnecessary; only quantities needed for immediate use need to be drilled out.

#### **B) CERAMIC MATERIALS**

Crucibles made of various kinds of ceramic materials can be used. The reader is referred to the text and tables in Part I, p. 12 ff., especially Table 7. Recently, crucibles of Ce and Th sulfides have proved advantageous for the fusion of nearly all metals, the exception being Pt. They can be used up to 1800°C [3].

Crucible shapes frequently used in the study of alloys are the conical (the so-called high shape) and the long, cylindrical (Tammann tubular crucible), both with a rounded or, less frequently, a flat bottom. Ceramic crucibles may be fitted with lids of the same material, but these can usually not provide a gas-tight seal by themselves (some sealing compound must be provided). Only alumina ( $Al_2O_3$ ) can be fused and then only in the case of very small tubular crucibles (about 15 mm. O.D.). In this case a well-fitting plug may be fused to the crucible wall with an acetylene-oxygen flame (welding torch) [4]. Thus, a crucible of the type illustrated in Fig. 350 is charged to a quarter of its



Fig. 350. Tubular alumina crucible. O.D. about 15 mm., length 65 mm.

height with the metal, closed off with the loosely fitting plug. and evacuated and filled with a suitable inert gas in the apparatus of Fig. 347. The crucible is then surrounded by moist sand to one half of its height, and the top section, including the plug, is carefully heated, using first a city gas-air mixture, then the acetylene-oxygen flame. The flange of the plug eventually fuses to the rim of the crucible (m.p. of  $A1_2O_3 = 2050$ °C). The entire closed crucible is then cooled carefully and uniformly. Some experience is necessary to avoid cracking during the fusion and, especially, during the cooling.

Tubular crucibles made of Pythagoras mass\* may be closed off in the same way as an ampoule, that is, by pulling to a small diameter and sealing the top end in a hydrogen-oxygen or acetylene-oxygen flame (see ampoule methods, p. 1782).

In another method, the crucible is lined with other materials. This method allows using, in the preparation of alloys, chemically resistant materials that cannot be shaped like a ceramic when unsupported.

Thus, CaO linings are suitable for work with Ca alloys and, in general, with calcium metal, which is extremely corrosive when hot. The Jander method of lining iron crucibles with CaO (already mentioned in Part I, p. 13) consists of the following.

The crucibles are 12 cm. high and 2.5 cm. I.D., with a wall 1 mm. thick. A thick paste of freshly prepared quicklime (from precipitated CaCO<sub>3</sub>) and water is partly poured in and partly painted on the inside of the crucible so that there is a layer 1-2cm. thick at the bottom and a wall lining 0.3-0.4 cm, thick. Rotating the crucible and careful pressing with a spatula produce a good and even coat. The crucible is then very slowly dried at room temperature or at 30-40°C; any cracks that appear are filled by pressing with a spatula (this can be done as long as the cake is

<sup>\*</sup>Pythagoras mass is a low-melting porcelain used for laboratory ware and electrical resistor casings, useful for temperatures not exceeding 1500 °C. Its melting point is about 1730 °C (Houben-Weyl, Allgemeine Laboratoriumspraxis [General Laboratory Practice], 4th ed., part 2, Georg Thieme Verlag, Stuttgart, 1959, p. 634).

still moist). Cracks that appear after this cannot be remedied; very fine cracks do no harm, as experiments with cracked CaO linings have shown. After the initial drying the temperature is increased to dark red heat, which transforms the Ca(OH)<sub>2</sub> to CaO. Dry CaO does not adhere well to iron. Since its coefficient of expansion differs from that of iron, the crucible must be heated very carefully and treated very gingerly even after the lining process. The crucible is charged with the reactants, and an iron lid, also coated with CaO, is welded on.

Coatings of LiF are suitable for work with lithium alloys and metallic lithium at temperatures below  $800^{\circ}$ C. These coatings adhere relatively well to zirconia ( $ZrO_2$ ) crucibles [6]. Thus, several grams of LiF (m.p.  $870^{\circ}$ C) are placed in a  $ZrO_2$  crucible which is positioned in a small, movable electric furnace. The crucible is firmly seated in the furnace by means of asbestos wool. A clear melt is produced on heating; the current is then shut off and the melted LiF evenly distributed over the crucible walls by tilting the furnace. This is continued until the crucible cools sufficiently for the material to set. Afterward, the furnace is allowed to cool slowly at a low current (from 700°C to 300°C in one hour). The lithium fluoride lining thus formed has a thickness of 1-2 mm. If the cooling is too rapid, it will have large cracks; small cracks are nearly unavoidable, but are not deleterious because of the high surface tension of most metals.

Other lining materials, such as Nucerite, which can be directly bonded to metals, under certain conditions withstands temperatures up to 650°C, and is resistant to many gases, are also on the market. In addition, such materials as Pyroceram will be quite useful in the high-temperature laboratory.

Of late, many new ceramic materials have been developed for use in the various military and space programs. It is not possible to list them in this short section. Besides, this field is undergoing very rapid changes and new materials appear almost monthly. The reader is therefore advised to spend some time consulting the pertinent trade literature before proceeding with the experiment. He may find such investment of time very worthwhile, because it may result in a simpler, better, more convenient and cheaper experimental arrangement.

Ceramic crucibles may also be placed in glass, quartz or ceramic combustion tubes (one end open) so that the material may be in a vacuum or an inert gas atmosphere during the heating. Such an arrangement is shown in Fig. 272, p. 984.

Ceramic crucibles are frequently enclosed in slightly larger iron crucibles, which are then hermetically sealed with a weldedon plug; this arrangement combines the chemical resistance of ceramics, especially the oxides of Be, Mg, Al, Zr and Th, with the ease of sealing of iron crucibles. Finally, all the crucible types mentioned above may be sealed into glass ampoules: this permits maintaining a vacuum or a desired gas atmosphere during the fusion.

### C) AMPOULES

All types of glass, especially the high-melting glasses (see Part I, p.5 ff.) and quartz, as well as tubes of Pythagoras mass, can be formed into ampoules (bomb tubes) and used for alloying of metals by fusion. The glass type used depends on the maximum working temperature. Pyrex can be used up to 560°C, Vycor up to 800°C under normal conditions and 1100°C for a short time, fused silica up to 1150°C, and Pythagoras mass up to 1400°C without danger of deforming. The metal reactants are changed into a long combustion tube of the appropriate ampoule material. The tube will normally have an I.D. of 10-20 mm., a wall thickness of 1.5-2 mm., and a round bottom of uniform thickness. The tube must be thoroughly cleaned and dried. It is then constricted just above the charge, but not so close to the latter that a reaction will be set off by the heat applied during sealing. The wall must be fairly thick at the constriction. The tube is evacuated and sealed at the constriction, thus forming an ampoule containing the metals under vacuum. The constricting and sealing are done with suitable torches (city gas-air,  $H_2$ - $O_2$ , etc.), depending on the softening temperature of the ampoule material.

The ampoule can also be filled with a protective gas. However, the thermal expansion of the gas must be taken into account in this case. For this reason, the ampoule is filled with the requisite gas at less than atmospheric pressure at room temperature.

After fusion and resolidification, the ampoule is broken up and the metal ingot isolated. The composition of the alloy may then be calculated (approximately) from the weights of the reactants and the product. However, only careful chemical analysis can give the true composition.

## Heating and Cooling

The required reaction or fusion temperature is determined from the phase diagram. As a minimum, this temperature must be higher than the liquidus point of the alloy product. Preferably, however, it should exceed the melting points of all the reacting metals. The best temperature is one which exceeds the liquidus point by 30-50 °C over the entire range of compositions of the system. Such a temperature will certainly ensure proper reaction conditions.

Heat sources may be furnaces of various types (see Part I, pp. 32-42). In general, the materials may be heated up as

rapidly as desired. The temperature increase due to the heat of reaction may be neglected. The heating time should be as short as possible to avoid reactions between the metal and walls of the reactor, and should in no case be longer than the time absolutely necessary to achieve a uniform composition. For this reason, the furnace should be preheated to approximately the desired temperature prior to the introduction of the vessel with the reactants. This vessel must, of course, be heated slowly enough to avoid stresses which would produce breakage. This applies particularly to glass ampoules and ceramics of low thermal conductivity. In any case, well-designed protective glasses or goggles must be worn during these operations.

When the desired melting temperature is reached, the homogenization of the mixture is promoted by mechanical means. Open crucibles are stirred with a rod of suitable material; tightly closed vessels (crucibles with a screwed-on or welded-on lid, ampoules) are taken out of the furnace and shaken or tumbled a few times; in the case of crucibles which are open but surrounded by a second protective vessel and which thus cannot be shaken or tumbled, at least some motion of the melt can be induced by external tapping or vibrating. All such agitation procedures must be followed by a short reheating to the maximum desired temperature.

**Cooling** also depends on the phase diagram as well as the intended use of the alloy. If there is no danger of separation of mixed crystals (with subsequent alteration of the composition of the alloy) and no peritectic reactions are expected, or if the composition achieved at the high temperature is the one desired in the solid, the material is quickly cooled in air. Materials in metal or quartz vessels may also be quenched in water or oil.

On the other hand, when a reaction must be completed at a lower temperature or it is desired to produce single crystals for studies on structure, then a slow, controlled reduction of the temperature is required. The type of cooling procedure thus depends on the application.

The formation of large single crystals from the melt may be favored by quiet, vibrationless cooling. Sometimes, however, motion of the melt during crystallization is desirable. The heating and cooling methods used for single crystals of pure metals are also applicable to single crystals of intermetallic compounds that exhibit congruent melting (see Part I, p. 94 ff., and [7]).

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# Alloy Synthesis under Pressure

Special methods are required when one of the constituent metals of the alloy has a very low boiling point (Zn, Cd, Hg; see also p. 1789) while the other constituent has a high melting point (platinum metals and other transition metals; see Table 4, p. 1776). In this case, if the pressure is atmospheric pressure, one metal tends to vaporize before the other liquefies. Nowotny et al. have designed a special furnace which allows heating such metal combinations in a protective gas at high pressures. The apparatus is essentially a closed iron bomb containing a resistance-heated tube which encloses the crucible (Fig. 351). The furnace mantle m is a thick-wall seamless steel tube whose lower section is threaded for 50 mm. and carries a screw cap v. This cap in turn carries a threaded adapter which is connected to the compressed gas cylinder, the pressure gage and one of the two electrical terminals. All screw connections are also sealed with lead gaskets b. The other end of mantle  $m_{\parallel}$  is closed off by cover plate d (a 35-mm.-thick circular steel plate) held in place by flange f and connected to the second electrical terminal. The gasket ring r is made of insulation-grade asbestos, and the six flange screws are of high-strength nickel-chromium steel (the screws must also be insulated from the flange by sleeves and washers of electrically insulating asbestos or similar material). The inner walls of plate d and the cap v are threaded so as to support the brass collars  $h_1$  and  $h_2$ . The latter make the electrical connection between the plate and cap and the carbon resistance element which they support. To obtain good electrical contact between the brass collars and the carbon sleeve conductors  $g_1$  and  $g_2$ , the I.D. of the collars is made 0.1 mm. smaller than the O.D. of the corresponding section of the carbon sleeve. The collar is then heated and slipped over the

sleeve while still hot. The heating tube k is made from electrographitized carbon and has a wall thickness of 1.5-1.8 mm. in the long middle section and 4 mm. at the ends. Its ends fit tightly into sleeves  $g_1$  and  $g_2$ .



Fig. 351. Tubular furnace for high temperature pressure synthesis. k carbon tube;  $g_1$ ,  $g_2$  special carbon sleeve conductors;  $h_1$ ,  $h_2$  brass collars; m outer steel tube (mantle); f flange; d cover plate; v screwedon cap; b lead gaskets; r, i asbestos gasketing rings and sleeves. Dimensions in mm.

The narrow, high tubular crucible containing the charge is centered in k. The furnace is connected to a low-voltage transformer capable of delivering 600-900 amp. at 12 volts. The apparatus is filled with a protective gas (N<sub>2</sub> or Ar) to 60-70 atm.; the pressure increases rapidly to about 150-200 atm. during heat-up but then decreases again during the actual fusion (it drops to about 70-100 atm. in 10 min.). The temperature cannot be measured directly;

G. BRAUER

it can only be estimated from the current consumption, assuming otherwise constant conditions (preliminary experiments are helpful). During the run, the furnace is immersed in running water up to the flange while the lid is cooled by spraying water from above.

Electrographitized carbon is a useful crucible material for preparation of alloys of Zn and Cd with Pt or Pd. Alumina crucibles do not last. The volatile-metal loss may approach 25%. The carbon tube and sleeves may last for 40-70 fusions (10 minutes each). Obviously, this furnace is not suitable for metals which readily form stable carbides.

REFERENCE:

H. Nowotny, E. Bauer and A. Stempfl. "Alfons-Leon-Gedenkschrift" der Allg. Bau-Zeitung, Vienna, 1951, p. 63.

## Melting Without a Container

Under certain conditions it is possible to melt small quantities of metals, alloys and related compounds in such a way that they do not make or barely make contact with the wall of the container. Such a procedure becomes very desirable when one deals with corrosive elements or when products of very high purity are required. However, "containerless" fusion is possible only in special cases. For example, the sample may be heated to melting by means of an electric arc or a directional electron beam; in this case the sample rests in a shallow depression in a cooled copper plate. The molten sample contracts due to surface tension to form an oblate spheroid whose area of contact with the copper support is so small that no contamination occurs during the short fusion process. The resolidified sample is turned over and remelted on the other side. This procedure is called button melting. For heat sources, see Part I, p. 42.

Another melting method is the so-called levitation melting in which the sample is freely suspended in vacuum or in an inert atmosphere by a field developed by means of induction coils, which also supply the heat. This promising method is, however, still in the experimental stage, [E. C. Ocress, D. M. Wroughton, G. Comnetz, P. H. Brace and J. C. R. Kelly, J. Appl. Phys. 23, 545 (1952); J. Electrochem. Soc. <u>99</u>, 205 (1952)].

## Comminution in the Absence of Air

Special precautions must be taken while studying alloys that are extremely sensitive to air, hygroscopic or readily oxidized. This

1786

is especially true of operations in which one is trying to obtain comminuted material, shavings, etc., for density determinations or x-ray powder diagrams, where such material must be completely free of decomposition products. Devices for producing such comminuted alloys have been developed by Zintl et al. (I) and Klemm and Dinkelacker (II).



Fig. 352. Comminution of sensitive alloys in the absence of air.  $k_1$ ,  $k_2$  working chambers for handling the alloy;  $h_1$ ,  $h_2$  high-vacuum stopcocks; *m* capillary for x-ray sample; *u* glass tube for annealing the alloy powder; *o* electric furnace; *a* adapter with the analysis and sample storage bulb; *f* rotary milling tool.

I. The apparatus in Fig. 352 consists of several glass parts connected by ground joints  $s_1-s_4$ . The assembly is connected via Schiff stopcocks  $h_1$  and  $h_2$  (see Part I, p. 59 f.) to a high-vacuum pump and a supply of pure, anhydrous inert gas.

The major constituent parts of the apparatus are two slightly oblate chambers  $k_1$  and  $k_2$  made of medium-wall glass tubing. The underside of each chamber carries three to four corrugations impressed with a carbon rod on the hot, soft tube. These corrugations prevent the alloy from slipping out of the chamber during workup. Depending on need, devices for annealing the comminuted material, for charging the glass capillaries which are used to hold the sample while examining its x-ray powder pattern, or for removal of analytical samples (analysis bulbs) may be added.

Before use, the entire apparatus is thoroughly evacuated over a period of several hours. At the same time, it is carefully heated (to promote removal of moisture) and then connected via stopcock  $h_1$  to the protective gas supply. The blanketing gas is allowed to escape at  $h_2$ . A small piece of the solid alloy is then introduced into chamber  $k_1$  via the ground joint  $s_4$ . A fast stream of the protective gas is maintained while introducing the sample, following which a loosely fitting rubber cap is put over joint  $s_4$ . This permits the gas to escape while preventing air penetration into  $k_2$ . The surface of the alloy in  $k_1$  is cleaned by means of a small rotary milling tool f (5 mm. O.D.), set on a 12-cm.-long rigid shaft and introduced via  $s_4$ . The rigid shaft is attached to the chuck of a flexible shaft driven by a dental drilling machine. The alloy piece is cleaned on all sides and is then pushed into the chamber  $k_2$  by means of a thick wire or a thin glass rod. The powdered waste material removed from the metal surface must not be entrained from  $k_1$  into  $k_2$  (it can be dislodged from the metal surface by tapping).

When the material is safely in  $k_2$ ,  $k_1$  is disconnected at  $s_3$ . The required quantity of clean shavings is then produced from the alloy in  $k_2$  (a new, clean milling tool should be used). After this,  $s_3$  is reconnected to  $h_2$ . Turning and tapping the assembly transfers the fresh alloy shavings (or powder) to tube u (this tube must be prebaked and degassed in high vacuum). The alloy powder in umay then be annealed in the heat produced by furnace o. This treatment removes stresses and is frequently necessary in order to obtain good powder patterns with sharp interference peaks. If very sensitive alloys are handled, the temperature maintained during degassing of tube u must be higher than during the succeeding annealing of the sample. After the annealing, the required quantity of powder is transferred to the capillary tubes m, which are then melt-sealed prior to introduction into the x-ray powder pattern analyzer (the wide end of tube m is cemented to the adapter at  $s_1$ , and a small side opening serves to equilibrate its pressure with that in the protective tube p).

If desired, tube u may be replaced by a bulb for sealing off analytical samples. The net weight of the empty bulb and its adapter (to  $s_2$ ) is first established. The bulb is then filled with alloy powder and sealed off. It is then reweighed, and the total weight of the oxide-free metal powder can thus be accurately determined.

II. The arrangement of Klemm and Dinkelacker also utilizes a small rotary milling tool for cleaning of the surface and comminution of the alloy chunk. In this method, however, the alloy is not removed from the crucible but is powdered while still in the crucible. The apparatus is shown in Fig. 353.

The thick-wall brass shell a houses vessel b, which can be rotated on axis t. The top of the crucible containing the freshly prepared alloy is sawed off, and the crucible is introduced via the ground-joint adapter c, which slopes upward. The crucible is then firmly fastened in b by means of a small screw r. Vessel b can be turned into any desired position by means of handle d. It is then fixed in that position by turning down screw s, which thus immobilizes axis t. Housing a can be evacuated via the groundmetal joint f, while additional vessels for further treatment of the alloy (analysis, powder pattern, density determination, etc.) can be

attached at a similar joint e. The apparatus can be completely sealed and evacuated if handle d is taken off and a standard taper cap pushed over the ground joint g.

The surface of the alloy in the crucible is cleaned by means of a small, rapidly rotating steel milling tool (5 mm. diameter), driven from a dental drill via a flexible shaft and introduced through joint c. The impurities removed from the surface of the alloy are dumped into e by turning b on axis t. A disposable rubber wiper blade w cleans off all waste powder from the walls of a and pushes it into e. The waste is then removed from e, the joint is cleaned by blowing through it inert gas, and a new cutting tool is introduced through c. The clean alloy shavings (see the previous method) thus produced are removed through  $e_{\bullet}$ 

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# **Distillation Method**

If one component of a desired binary alloy is more volatile than the other and if the decomposition vapor pressure of the alloy is not too high, the alloy can be prepared by distilling or subliming the volatile component onto the other.

However, apart from a few exceptional cases this method is, for obvious technical reasons, restricted to alloys made of relatively volatile metals and metalloids, that is, those boiling below 1000-1100°C at 760 mm. This is because distillation at higher temperatures is quite difficult in practice. (However, much less volatile



Fig. 353. Comminution of alloys in the absence of air without removing the alloy from the crucible. a brass housing; b vessel housing the crucible and able to rotate; t axis on which b rotates; d handle; wrubber wiper blade for scraping off loose particles.
metals can be distilled in small amounts in high vacuum.) The method is therefore suitable for:

Na, b.p. 880°C,	Mg, b.p. 1107°C,	P, b.p. 280°C,
K, b.p. 760°C,	Zn, b.p. 907°C,	As, b.p. 615°C,
Rb, b.p. 700°C,	Cd, b.p. 767°C,	Se, b.p. 688°C.
Cs, b.p. 670°C,	Hg, b.p. 357°C,	

The special advantages of the distillation method are as follows:

a) The volatile component is repurified by the distillation just prior to the reaction (this is important in the case of very reactive metals).

b) The reaction between the vapor of one component and the powder of the other proceeds quietly and smoothly (because of the limited amount of vapor present at any time).

c) Any excess of the volatile component can be distilled off after the reaction.

Each of the components is placed in a separate boat and the boats are positioned one behind the other in a horizontal tube. The choice of boat and tube materials is governed by the same considerations of thermal stability and chemical resistance as were discussed in the case of crucibles (see p. 1775 ff.). The tube must be gas-tight. For this reason, it is usually closed at one end and carries a ground joint on the other (the latter is for evacuation and filling with inert gas); alternatively, it may carry high-vacuum valves or stopcocks on both ends (see Fig. 354). To protect the glass, quartz or ceramic tube a against corrosion by the volatile metal, a liner tube b (made of glass, ceramic material or metal such as Fe or Ni) may be inserted. Boat  $s_1$  contains an excess of the volatile component, while boat  $s_2$  is filled with the finest possible powder of the relatively nonvolatile reagent. Boat s<sub>1</sub> is also surrounded by a test-tubelike cylinder c which acts as a vapor deflector.

At the start of the run, a high vacuum is created in tube a. Then the two short, tubular electric heaters are switched on and regulated in such a way that a temperature sufficiently high to maintain a reasonable rate of distillation exists in  $s_1$ , while a slightly lower temperature exists in  $s_2$ . The temperature in  $s_2$  should be sufficiently high to induce and maintain the reaction between the metal powder and the vapor arriving from  $s_1$ . At the end of the run, the excess of the volatile component is distilled off and condensed in the cooler section of the tube. Finally, the product is removed from boat  $s_2$ .

The above method is also useful for purifying a crude product obtained from two components by the crucible or ampoule fusion methods. The excess of the volatile component may thus be removed by vacuum distillation. In this case the vapor pressure of the free volatile component must, of course, be much higher than its pressure in the residual intermetallic phase. Examples of application of this method are preparations of silicides and germanides of alkali metals, and of  $Na_3As$  and  $K_3Sb$ .



Fig. 354. Preparation of alloys by distillation. *a* reactor tube; *b* liner tube; *c* vapor deflecting cylinder;  $s_1$ ,  $s_2$  boats;  $o_1$ ,  $o_2$ tubular electric heaters.

The same principle can also be applied to the Faraday sealedtube system described in Part I, p. 76 f. This system is completely closed and the only external influence consists of the temperature gradient; direct handling is not possible in this case. However, if the reactor tube material is resistant to corrosion by the reagents involved, the Faraday system produces an extremely pure reaction environment.

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## **Residue Methods**

Occasionally, a pure component ( $\alpha$ ) phase may exhibit properties markedly different from those of the intermetallic phase which is vicinal to it on the constitutional diagram. Thus, the  $\alpha$ -phase may dissolve more readily in a solvent or it may be attacked more readily by a reagent. In such cases it may be possible to use an excess of the pure component during the high-temperature synthesis and then liberate the intermetallic product by leaching out the matrix phase. Occasionally, also, slow cooling of the alloy melt may yield well-formed crystals of the intermetallic phase embedded in the pure component matrix, which may then be removed by some solvent. Depending on circumstances, the matrix phase may be removed by electrolytic oxidation, by aqueous acids, by bases, or by liquid NH<sub>3</sub>. For example:

I. Electrolytic solution processes may be used to isolate semimetallic compounds from the matrix in which they are produced. Thus, for instance,  $Fe_3C$  may be isolated by electrolytic oxidation of the surrounding carbon-rich steel (see p. 1503). Other carbides such as (Fe, Cr)<sub>3</sub>C, (Cr, Fe)<sub>7</sub>C<sub>3</sub>, (Fe, Mn)<sub>3</sub>C, (V, Fe)<sub>4</sub>C<sub>3</sub> and  $Fe_3Mo_3C$  are prepared in a similar manner. A procedure and an



Fig. 355. Extraction of alkali metal alloys with liquid ammonia.

a similar manner. A procedure and an especially elegant apparatus have been developed by Klinger and Koch. This procedure has been employed primarily for the study of steels containing nonmetallic admixtures, but should be usable for a more general study of alloys.

II. Aluminum compounds such as  $Al_3Ti$ ,  $Al_3Zr$ ,  $Al_3Th$ ,  $Al_3V$ ,  $Al_3Nb$ ,  $Al_3Ta$ ,  $Al_4Ce$ ,  $Al_4La$ ,  $AlB_2$ , etc., may be isolated from the Al-rich (matrix) products of the corresponding aluminothermic reactions, the solvents being dilute acids or bases. In a similar way, some silicides such as  $ZrSi_2$ ,  $ThSi_2$ ,  $VSi_2$ ,  $NbSi_2$ ,  $TaSi_2$ ,  $MoSi_2$ ,  $WSi_2$  and  $USi_2$  may be produced in molten aluminum "solvent" and then isolated as residues from treatment of the respective aluminum alloys with acids or bases (see p. 1794).

III. A special case is the isolation of intermetallic compounds of alkali or

alkaline earth metals (except Be and Mg) by extraction with liquid ammonia. The alkali or alkaline earth metal, in considerable excess, is fused with the alloying component, the mixture is slowly cooled to obtain large crystals, and the solidified melt is then transferred without being exposed to air (the apparatus in Fig. 349 is quite appropriate here) into an extraction apparatus such as the one shown in Fig. 355. This device was developed on the basis of the arrangement of Biltz and Rahlfs (see Part I, Fig. 71). The alloy to be extracted is introduced through a and placed on top of the dense glass-wool filter f. The stopcocks  $h_1$  and  $h_2$  are both connected to a single vessel so that one can establish a high vacuum in the system, introduce NH<sub>3</sub> to one (or both) tubes, or establish a connection between tubes b and c. Some NH<sub>3</sub> is condensed in b, where it contacts the alloy on f; the blue solution of the alkali metal is then allowed to pass via the interconnecting tube d into tube c. There NH<sub>3</sub> is evaporated, leaving behind the free alkali metal. The  $NH_3$  vapor is recycled to b for recondensation. Several repetitions of this operation allow exhaustive extraction of the alloy on f. The desired compound, which is completely resistant to or attacked only slightly by the  $NH_3$ , remains as a residue on  $f_*$ .

The apparatus shown in Part I, Fig. 73 may also be used for the extraction of alloys with liquid ammonia.

The above method was used to produce  $Na_3As$ ,  $Na_3Sb$ ,  $Na_3Bi$ ,  $Na_{15}Pb_4$ ,  $Na_{15}Sn_4$ ,  $NaZn_{13}$  and  $Na_2Au$ . Pure, well formed single crystals were obtained.

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## **Special Processes**

Intermetallic and metalloid compounds may also be prepared by methods other than those described in the preceding sections. However, these other processes have so far been used only in special cases, since the necessary conditions tend to limit their general applicability.

# I. SIMULTANEOUS CHEMICAL REDUCTION OF NONMETALLIC COMPOUNDS AND ALLOYING OF NASCENT FREE METALS

For instance, reduction of niobium oxide and nickel mixtures by means of hydrogen leads to Ni-Nb alloys [1]. Thermal decomposition of isomorphous mixtures of Fe, Co or Ni formates or oxalates, conducted under reducing conditions, gives fine, crystalline alloy powders. These alloys correspond to phase equilibria at comparatively low temperatures [2].

## II. SYNTHESIS OF BINARY ALLOYS AND INTERMETALLIC COMPOUNDS BY COMBINING SOLUTIONS OF BOTH COMPONENTS

The solvent may be a third, more or less inert metal, usually with a low melting point (Hg, Al or Mg). The process temperatures

thus range between ambient and several hundred degrees. For basic data on the precipitation of intermetallic compounds, see [3].

The Hg solution method may be used for the preparation of many intermetallic compounds, as well as very reactive metals. This is because the Hg solvent can be distilled off at a comparatively low temperature. The method can thus be used for preparation of alloys which cannot be obtained by fusing or sintering at high temperatures (see Amalgam Metallurgy [4]).

Aluminum compounds (silicides, borides, and so forth) can be prepared in liquid Al (see p. 1797 ff.). Special silicides may be obtained in liquid Cu (see p. 1796).

Liquid  $NH_3$  may also be used as a solvent, especially in the synthesis of alloys of alkali and alkaline earth metals. However, this method has so far been used mostly for nonmetallic or metalloid alkali compounds [5].

## III. ELECTROLYTIC DEPOSITION OF ALLOYS FROM AQUEOUS SOLUTIONS.

The composition of the alloy depends on the composition of the electrolyte, the reaction conditions, and special additives which favor the precipitation. Just as in the case of solidification of melts, alloys precipitated by the electrolytic method consist of heterogeneous crystallizates, solid solutions, or some intermediate phases. They may differ from the alloys produced at high temperatures. The differences may show up in phase boundaries and in some physical and engineering properties [6].

The following binary alloy systems have so far been prepared in this way: Cu-Zn, Cu-Sb, Cu-Bi, Cu-Pb, Ag-Zn, Ag-Cd, Ag-Au, Ag-Bi, Ag-Pb, Au-Cu, Au-Ni, Ni-Zn, Ni-Cd, Ni-Fe, Zn-Cd, Pb-Sn, W-Ni, W-Co and W-Fe.

Some intermetallic or metalloid compounds may also be obtained by high-temperature electrolysis of liquid melts of the corresponding metal compounds. Secondary reactions sometimes play an important role in this case. This method, developed mainly by Andrieux and Dodero [7], has so far been used for borides (see p. 1798), silicides (see p. 1796 f.), phosphides, arsenides and carbides.

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## Silicides

A summary of processes for the preparation of metal silicides is given in Table 5.

I. MOISSAN'S CLASSICAL PROCESS (FUSION OF THE ELEMENTS)

These reactions are usually highly exothermic; the charge thus heats up far above its melting point, and a closely controlled reaction becomes impossible due to interaction with the walls of the vessel and gases, as well as volatilization of reagents; product purity and the yield are usually poor.

a) Silicides of transition metals, especially those of metals of Groups IV to VII, may be successfully prepared by sintering mixtures of powders of the constituents at comparatively low temperatures (< 1500°C). In this case, external heating must be discontinued promptly at the beginning of the exothermic reaction to avoid melting of the charge. The starting materials must be extremely pure and have a particle size < 0.06 mm. The powder mixtures are either tableted or pressed into alumina or graphite crucibles. They are heated in argon or in vacuum. Under these conditions, side reactions with the crucible materials are usually negligible.

#### G. BRAUER

#### Table 5

#### Preparation of Metal Silicides

_	Process	Reactions involved
Ι.	Synthesis from the elements	
	a) By fusion	$M + Si \rightarrow MSi$ , $MH + Si \rightarrow MSi + H_{2}$
	b) By sintering or sin- tering under pres- sure	
II.	Reaction of metal ox- ides with Si or $SiO_2$ (silicates) and C	$ \begin{array}{l} MO + Si \rightarrow MSi + SiO \\ MO + SiO_2 + C \rightarrow MSi + CO \\ M \ silicate + C \rightarrow MSi + CO \\ (unfavorable: \\ MO + Si \rightarrow MSi + SiO_2 \\ MO + SiC \rightarrow MSi + CO \end{array} $
III.	a) Aluminothermic and magnesothermic processes	MO+A1(Mg)+SiO <sub>2</sub> +S→ MSi (in A1)+A1(Mg)-S-containing slags
	b) Aluminum silicide process	$\begin{array}{l} \mathrm{Al-Si} + \mathrm{M} \rightarrow \mathrm{MSi} \ (\mathrm{in} \ \mathrm{Al}) \\ \mathrm{Al-Si} + \mathrm{MF}_{2} \rightarrow \mathrm{MSi} \ (\mathrm{in} \ \mathrm{Al}) + \mathrm{AlF}_{3} \\ \mathrm{Al-Si} + \mathrm{MO} + \mathrm{NaF} \rightarrow \\ \mathrm{MSi} \ (\mathrm{in} \ \mathrm{Al}) + \mathrm{Na}_{3} \mathrm{AlF}_{6} + \mathrm{Al}_{2} \mathrm{O}_{3} \end{array}$
	c) Copper silicide process	Cu-Si + M → MSi (in Cu) Cu-Si + MO → MSi (in Cu) + CuO • SiO <sub>2</sub>
IV.	Electrolysis of a melt	K₂SiF <sub>6</sub> +MO→MSi+KF
v.	Vapor-deposition proc- ess	$M + SiCl_4 + H_2 \rightarrow MSi + HCl$

b) According to Kieffer and Cerwenka, the density of the material can be increased even during the heating process, the result being a better product. A Tammann furnace (see Part I, p. 39) is used. The powder mixture is pressed into strong, 15-mm.-I.D. graphite molds and heated at 200 kg./cm.<sup>2</sup> and 1100-1500°C. After cooling, the surfaces of the samples thus obtained are ground, yielding a material containing only 0.02-0.05% C.

II. REDUCTION OF METAL OXIDES WITH Si, SiC OR SiO<sub>2</sub> (SILICATES) IN THE PRESENCE OF CARBON

In general, this process requires very high temperatures and yields fused products from which it is difficult to isolate well-defined

silicides. For this reason, this method is largely of historical interest. However, a modern variant is of some importance. In this variant, very pure Si is added to the metal oxide in the stoichiometric ratio. If the stoichiometric ratio is maintained exactly, all of the oxygen from the metal oxide will be bound to the Si, which then volatilizes as SiO. This method, which requires a vacuum but only relatively modest temperatures, yields very pure silicides.

The above process is applicable in all those cases in which both the metal and its oxide (which is reduced) have low vapor pressures at the reaction temperature (this is true of transition metals of Groups III to VIII). For instance, the process yields pure rare earth silicides, which are otherwise difficult to obtain. In this case the optimum reaction pressure is approximately 0.1 mm.

## III. ALUMINOTHERMIC OR MAGNESOTHERMIC PROCESSES

The general method was invented by Hönigschmidt. It starts from metal oxides and  $SiO_2$  and gives pure products if the nascent silicide is embedded in an excess of the reducing metal (preferably Al). This is achieved by using an excess of  $SiO_2$  and of the embedding Al in the reaction mixture. Of course, the embedding aluminum metal becomes alloyed to some extent with the Si and the other metal of the mixture. The object of the embedding process is to form an ingot or nugget which can be easily separated from the surrounding nonmetallic slag. This separation is facilitated by the addition of fluxes (CaF<sub>2</sub>, cryolite or CaO) to the reaction mixture (Al<sub>2</sub>S<sub>3</sub>, which was used by Hönigschmidt, is not recommended). The silicide can then be isolated from the nugget by reaction with dilute acid or alkali. It is obtained in the form of a crystalline powder.

Some processes for silicides start from the metal itself rather than its oxide. The metal is thus reacted with Si in the presence of a melt of a third metal which serves as the solvent. Aluminum is usually the optimum solvent. Another method employs components initially prealloyed with Al (for instance, ThSi<sub>2</sub> is made from Th-Al and Si-Al alloys). Still another process proceeds by stages, whereby the components of the silicide are prepared *in situ* from other compounds, primarily a metal fluoride or oxide and an alkali fluoride. The nascent components then form the silicide.

In all the above variants of the basic process, the silicide is always embedded in excess Al. The process has been used for NbSi<sub>2</sub>, TaSi<sub>2</sub>, ThSi<sub>2</sub>, MoSi<sub>2</sub> and WSi<sub>2</sub>.

The Lebeau process, which uses copper as a solvent and an Si carrier, is no longer of any importance.

Alternate methods:

## IV. ELECTROLYSIS

Silicides of Ti, Zr and Cr as well as those of the rare earth metals may be obtained by electrolysis of melts of the metals with silicates or fluorosilicates of suitable composition.

## V. VAPOR DEPOSITION

Silicide layers may be obtained from  $H_2$ -SiCl<sub>4</sub> mixtures by deposition of an incandescent filament.

Procedures for the preparation of silicides are also found in other sections of this book (see Alkali Silicides p. 989 f., magnesium silicide p. 921 f., calcium silicide p. 946 f., silicides of Ti, Zr and Th, p. 1249 f.).

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## Borides

Table 6 lists the most important processes for the preparation of borides, especially those of the transition metals.

## Table 6

## Preparation of Metal Borides

	Process	Reactions involved
I.	Synthesis from the elements	
	a) Fusion	$M + B \rightarrow MB$ , $MH + B \rightarrow MB + H_2$
	b) Sintering or sinter- ing under pressure	
II.	Aluminothermic and magnesothermic processes	MO + B <sub>2</sub> O <sub>3</sub> + Al(Mg) → MB + Al(Mg) oxide
III.	Reduction of oxides with carbon	$MO + B_2O_3 + C \rightarrow MB + CO$
IV.	Boron carbide process	$\begin{array}{c} \text{MO(M, MH)} + B_4\text{C}(\text{C, B}_2\text{O}_3) \rightarrow \\ \text{MB} + \text{CO} \end{array}$
v.	Electrolysis of a melt	MO+alkali (alkaline earth) borate+ alkali (alkaline earth) fluoride → MB+alkali (alkaline earth) borate- fluoride
VI.	Vapor-deposition processes	M (M halide) + B halide + H₂ → MB + hydrogen halide

I. Synthesis via fusion of the elements entails such high heats of formation that the reaction temperatures become very high. As a result, there is interaction with the material of the vessel and the product boride becomes contaminated. On the other hand, all borides may be prepared by sintering the appropriate metal with amorphous boron powder, which should be as pure as possible (commercial grades now available contain 97-99% B). The reaction mixtures should be heated in alumina crucibles (W or Mo crucibles or boats may also be used) in vacuum (an argon atmosphere is occasionally also used). The reaction, which is always exothermic, starts at temperatures of  $700-1200^{\circ}$ C; the highest temperature may lie above  $2000^{\circ}$ C. In some cases, sintering under pressure in carbon tubes (mentioned as a possible method of synthesis for silicides—see p. 1796) can be used.

II. Reduction with Al or Mg allows the use of oxides as starting materials and eliminates the preparation of pure boron. On the

other hand, this method involves inconvenience of chemical separation of the product boride crystals from by-products. This separation becomes especially difficult in the case of products of the aluminothermic preparation of high-melting borides (those of transition metals of Groups III to VI). No solid ingot or nugget is formed in this case; the fine boride powder is occluded in  $Al_2O_3$  slag. According to Andrieux and Peffen, an excess of  $B_2O_3$ , CaO and Na<sub>2</sub>O should be added to the reaction mixture. The alumina slag thus becomes soluble in acids and may be separated more easily from the boride powder.

III. Originally the preparation of borides of metals such as Ti and Zr from  $B_4C$  involved a reaction in hydrogen:

 $\mathbf{or}$ 

$$7 \operatorname{Ti} + 3 \operatorname{B}_4 \operatorname{C} + \operatorname{B}_2 \operatorname{O}_3 = 7 \operatorname{Ti} \operatorname{B}_2 + 3 \operatorname{CO}$$
  
 $3 \operatorname{Ti} + 2 \operatorname{B}_4 \operatorname{C} + \operatorname{Ti} \operatorname{O}_2 = 4 \operatorname{Ti} \operatorname{B}_2 + 2 \operatorname{CO}$ 

Recently, however, a simpler process has been devised:

$$2 \, TiO_2 \ + \ B_4C \ + \ 3 \, C \ = \ 2 \, TiB_2 \ + \ 4 \, CO$$

The mixture of starting materials is pressed into pellets and heated in a tubular carbon furnace under high vacuum. Maximum temperatures of 1400-1900°C are required to produce the metal boride within a reasonable time. The method has been tested for borides of Ti, Zr, V, Nb, Ta and W.

Alternate methods:

IV. Small quantities of pure boron compounds can be prepared by electrolysis of melts. Boron, which is evolved on the cathode from alkali and alkaline earth borates, combines with the simultaneously precipitated metal.

V. Passage of gaseous mixtures consisting of a metal halide,  $BBr_3$  and  $H_2$  over incandescent carrier metals on the average yields only boride layers inhomogeneous, and solid products.

**REFERENCES:** 

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   Z. anorg. Chem. <u>259</u>, 1 (1949); L. Brewer, D. L. Sawyer,

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D. H. Templeton and C. H. Dauben. J. Amer. Ceram. Soc. <u>34</u>, 173 (1951); H. Nowotny, F. Benesovsky and R. Kieffer. Z. Metallkunde <u>50</u>, 258, 417 (1959).

- II. E. Wedekind. Ber. dtsch. chem. Ges. <u>46</u>, 1198 (1913); J. T. Norton, H. Blumenthal and S. J. Sindeband. Trans. AIME <u>185</u>, 749 (1949); S. J. Sindeband. Ibid. <u>185</u>, 198 (1949); J.-L. Andrieux and R. Peffen. French Patent 1,123,861 (1956).
- III. P. M. McKenna. Ind. Eng. Chem. 28, 767 (1936).
- IV. R. Kieffer, F. Benesovsky and E. R. Honak. Z. anorg. allg. Chem. <u>268</u>, 191 (1952); G. A. Meyerson and G. V. Samsonov. Zh. Prikl. Khimii <u>27</u>, 1115 (1954); C. T. Baroch and T. E. Evans. J. Metals <u>7</u>, 908 (1955).
- V. L. Andrieux. Thesis, Univ. of Paris, 1929; Rev. Métallurg. <u>45</u>, 49 (1948); G. Weiss. Thesis, Univ. of Grenoble, 1946; Ann. Chimie <u>1</u>, 446 (1946); J. T. Norton, H. Blumenthal and S. J. Sindeband. Trans. AIME 185, 749 (1949).
- VI. M. Moers. Z. anorg. Chem. 198, 243 (1946); J. E. Campbell, C. F. Powell, D. Nowicki and B. W. Gonser. J. Electrochem. Soc. <u>96</u>, 318 (1949).

## Amalgams

While some low-melting (< 1000 °C) metals fail to form alloys with mercury, the majority may be amalgamated simply by heating

### Table 7

## Solubility of Metals in Mercury at about 20°C (after Jangg and Bach)

Metal	Solubility in weight %	Metal	Solubility in weight %	Metal	Solubility in weight %
Aσ	0.03	In	57	Bu	0.35
AÎ	$2.3 \cdot 10^{-3}$	ĸ	0.38	Sb	$2.9 \cdot 10^{-5}$
As	0.24	La	0.013	Si	$1 \cdot 10^{-3}$
Au	0.13	Li	0.036	Sn	0.87
Ba	0.33	Mg	0.31	Sr	1.0
Be	$1 \cdot 10^{-3}$	Mn	$1.7 \cdot 10^{-3}$	Та	0.01
Bi	1.4	Mo	$< 2 \cdot 10^{-6}$	Th	0.016
Ca	0.3	Na	0.62	Ti	$1 \cdot 10^{-5}$
Cd	~ 5.0	Ni	$2 \cdot 10^{-6}$	U	$1 \cdot 10^{-5}$
Co	$< 1 \cdot 10^{-6}$	Pb	1.5	v	$1 \cdot 10^{-5}$
Cr	$< 4 \cdot 10^{-7}$	Pd	0.06	W	$1 \cdot 10^{-5}$
Cu	$2 \cdot 10^{-3}$	Pt	$2 \cdot 10^{-3}$	Zn	2.0
Fe	$< 5 \cdot 10^{-7}$	Rb	1.4	Zr	$3 \cdot 10^{-3}$
Ge	$1 \cdot 10^{-3}$	Rh	0.16		

with mercury in closed iron crucibles or glass ampoules. Under special circumstances, these vessels may also contain special insert crucibles (see p. 1774 ff.). This simple process applies to all permissible amalgam compositions.

Another general method for preparation of amalgams consists of the electrolysis of solutions of the respective metal salts in cells comprising an Hg cathode. The concentrations of the respective metals in the amalgam reach very high levels so that solid phases may separate. The method is applicable even to metals with extremely low solubility in Hg (for example, Fe); in this case, the method gives suspensions of the metal in Hg which exhibit behavior very similar to that of "true" amalgams (see Table 7).

REFERENCE:

G. Jangg and H. Bach. Quecksilber- und Amalgammetallurgie, Handbuch d. techn. Elektrochemie [Mercury and Amalgam Metallurgy, Handbook of Engineering Electrochemistry], Vol. I, Leipzig, 1961.

It is sometimes desired to prepare liquid or semisolid amalgams of base metals for special purposes, especially for use as laboratory reducing agents. The following simple methods apply in these cases:

## SODIUM AMALGAM (LIQUID; ABOUT 1% Na)

I. Clean sodium metal (11.5 g.) is cut into 5-mm. cubes. The cubes are speared with a pointed glass rod and rapidly introduced below the surface of warm ( $30-40^{\circ}$ C) pure mercury (1150 g. = 85 ml.) contained in a 500-ml. wide-neck Erlenmeyer flask. The flask is covered with cardboard to prevent spattering during the rather vigorous amalgam formation reaction.

II. In another method, 3.5 g. of Na protected by a layer of toluene (10-15 ml.) is melted in a 250-ml. Erlenmeyer flask placed on a hot plate. Then, 340 g. (25 ml.) of Hg is added in drops (stirring or shaking). The first few drops of mercury cause a vigorous reaction, but then the amalgam formation becomes less violent. The toluene boils during the entire addition; at the end, it is decanted or displaced with other liquids.

The following method for solid amalgams may be modified and adapted to other ratios of reagents.

## SODIUM AMALGAM (SOLID; ABOUT 2-3% Na)

I. The Fieser procedure gives especially pure material: Clean sodium pieces (6.9 g. for an amalgam containing 2% Na, 10 g. for one containing 3% Na) are placed in a 250-ml. three-neck

round-bottom flask. The two side necks carry nitrogen inlet and outlet tubes, while the center neck carries a dropping funnel containing 340 g. (25 ml.) of Hg. The flask is thoroughly flushed with  $N_2$ , and 10 ml. of Hg is then added. The flask is heated on an open flame until the start of the reaction. Additional Hg is then slowly added, with minimum additional heating. After the addition, the hot molten amalgam is poured onto a clean plate and broken up into pieces while still hot and brittle.

II. Another method is useful for preparation of larger quantities of amalgam. Thus, for example, 51 g. of clean, freshly cut sodium is heated in an enameled pot (about 18 cm. I.D.) under paraffin oil until molten (the thickness of the protective oil layer should be 1 cm.). Then, 1650 g. (122 ml.) of Hg is added slowly (constant stirring) from a dropping funnel. This reaction ends within 3-4 minutes, and most of the paraffin is decanted. The amalgam solidifies at about  $250^{\circ}$ C. During cooling, it is comminuted with a heavy pestle to form small beads. Alternatively, the hot liquid amalgam may be poured (together with the adhering oil) into a porcelain dish and allowed to solidify; the cake is then broken up to the desired particle size in a mortar. After complete cooling the oil is removed by washing with petroleum ether or benzene. The solvent is then evaporated, and the product stored in an air-free atmosphere.

PROPERTIES:

Amalgams containing less than 3% Na are not too sensitive to air; however they must be stored in an air-free atmosphere. Complete liquefaction occurs at the following (liquidus) temperatures: 0.5% Na, 0°C; 1.0% Na, 50°C; 1.5% Na, 100°C; 2.0% Na, 130°C; 2.5% Na, 156°C; 3% Na, 250°C; 4.0% Na, 320°C.

**REFERENCES:** 

Organic Syntheses, Collective Volume 1, New York-London, 1941,
p. 539; W. R. Renfrow, Jr., and C. R. Hauser in: Organic Syntheses, Collective Volume 2, New York-London, 1950, p. 609; V. Deulofeu and T. H. Guerrero in: Organic Syntheses, Vol. 22, New York-London, 1947, p. 92; S. H. Babcock in:
H. S. Booth. Inorganic Syntheses, Vol. 1, New York-London, 1939, p. 10; L. F. Fieser. Experiments in Organic Chemistry, New York, 1941, p. 419,

#### POTASSIUM AMALGAM

For preparation, see A. Roeder and W. Morawietz, Z. Elektrochem. <u>60</u>, 431 (1956).

#### CALCIUM AMALGAM

The reactor is a small steel bomb b (Fig. 356). The vertical part of the top opening carries a flat thread into which is screwed the steel head k. While the two parts should fit each other well, one should be to unscrew k without too much difficulty. The conical extended surface above the thread provides additional sealing area. The seal should hold at better than 70 atm. The head k carries a pressure-reducing value r connected to pressure gage m and pressure tubing d, which leads to a compressed nitrogen cylinder.



Fig. 356. Preparation of calcium amalgam. b steel bomb, 19 cm. long, 5.5 cm. I.D., 9 mm. wall thickness; capacity 45 ml. up to the screw thread; smovable piston; k steel head; r pressurereducing valve; m pressure gage; d copper pressure tubing for in-

troduction of N<sub>2</sub>.

A piston s, located inside the bomb and actuated by the gas from the cylinder, fits the walls fairly tightly.

To start with, the bomb is charged with 35 ml. of pure Hg. Then, 7-10 g. of shavings of commercial Ca are added and the piston set in place. The bomb is tightly closed and nitrogen injected to a pressure of 60-70 atm, above the piston. The formation of the amalgam starts immediately, heat of reaction is evolved, and the bottom of the bomb becomes hot; the reaction ends in no more than 10-15 minutes. The pressure is released and the bomb is opened. The amalgam is rapidly transferred to a dry, wide-neck flask of about 60-ml. capacity, which is then tightly closed. The steel bomb is rinsed with some pure Hg. This material is added to the flask, which is then filled with Hg to just below the stopper (the amalgam in the flask heats up considerably on dilution with the Hg). The mixture is shaken well and allowed to cool.

#### PROPERTIES:

This amalgam contains about 1% Ca; sometimes solidifies in the cold; can be stored indefinitely in containers com-

pletely filled with mercury. According to data in M. Hansen and K. Anderko, Constitution of Binary Alloys, the liquidus curve in the Ca-Hg system rises to about 260°C at 2% Ca, 140°C at 1% Ca, and 25°C at 0.3% Ca; most of this amalgam solidifies at the peritectic point of -39 °C.

REFERENCE:

### A. Brukl. Angew. Chem. <u>52</u>, 151 (1939).

#### STRONTIUM AND BARIUM AMALGAMS

Strontium and barium amalgams are prepared by electrolysis of solutions of the corresponding chlorides on mercury cathodes. The directions for the compound are each applicable to the other.

#### STRONTIUM AMALGAM

The cell a of the apparatus shown in Fig. 357 contains about 300 ml. of a saturated solution of pure SrCl<sub>2</sub>, made weakly acidic

with HCl (if contaminated with traces of Na and Fe the salt must be prepurified). The cathode b consists of 600 g. (45 ml.) of pure, distilled Hg. The thick Pt wire c, which dips into the mercury pool, is the current lead. The anode is a graphite rod e (10 mm. O.D.) suspended in a porous clay cell d which dips into the liquid. The electrolysis proceeds at a current of 6.5 amp. (that is, at 1.7 amp./in.<sup>2</sup>, considering the 24 cm.<sup>2</sup> of cathode surface). The passage of the current causes a sharp temperature rise and the electrolyte is maintained at the optimum temperature of 38-40°C by means of cooling coils g and f wound on the outer surface of cell a and immersed in the solution. The liquid in the anode space is replaced every 15 minutes to prevent accumulation of Cl<sub>2</sub> and its penetration into the cathode space. The Sr content in the Hg reaches 1.3% in 90 minutes: at this point the amalgam becomes a slurry because



Fig. 357. Preparation of strontium amalgam. *a* electrolysis cell; *b* mercury cathode; *c* current lead (Pt wire); *d* porous clay cell; *e* anode (graphite rod); *f*, *g* water-cooled coils.

the liquidus point for this composition is reached. The current is shut off and the electrolyte is poured out from the cell. The amalgam is then removed, washed several times with water, and dried with filter paper.

#### **PROPERTIES:**

Silvery, shiny, solid at room temperature. Must be stored in air-free atmosphere.

REFERENCE:

L. Holleck and W. Noddack. Angew. Chem. 50, 819 (1937).

BARIUM AMALGAM (APPLICABLE ALSO TO STRONTIUM AMALGAM)

The cell consists of a 250-ml. beaker. The cathode is a pool of 250 g. (18 ml.) of pure mercury. A Pt wire, all of it except for the tip sealed in glass to prevent contact with the electrolyte, dips into the mercury pool and serves as the current lead. The anode is a horizontal sheet of Pt,  $5-10 \text{ cm.}^2$  in area.

The cell is filled with 100 ml. of saturated  $BaCl_2$  solution. The electrolysis proceeds at a current of 1.75-2.5 amp. and 6-7 v. The current should be shut off after 2.5 hours, since beyond that time there occurs a sharp voltage rise, evolution of gas at the cathode and decomposition of the amalgam. Sometimes, crystalline amalgam particles deposit on the mercury surface and interfere with the electrolysis. This deposition may be prevented by slow agitation with a stirrer, or the crystals may be pushed into the body of the mercury with a glass rod.

At the end of the run the solution is decanted and the amalgam is thoroughly washed with distilled water, followed by ethanol and ether. It contains about 3% Ba. Is is stored in an air-free atmosphere.

**REFERENCES:** 

G. McPhail Smith and A. C. Bennett. J. Amer. Chem. Soc. <u>31</u>, 804 (1909);
B. C. Marklein, D. H. West and L. F. Audrieth in: H. S. Booth. Inorg. Syntheses, Vol. 1, New York-London, 1939, p. 11.

ZINC, CADMIUM, TIN, LEAD AND BISMUTH AMALGAMS (LIQUID)

The liquid zinc amalgam (2-3% Zn) is prepared from 4 g. of zinc (granules, shavings; preferably, however, foil). The zinc is degreased with ether, thoroughly washed with dilute H<sub>2</sub>SO<sub>4</sub>, placed in a 100 ml. flask, and heated on a water bath together with 200 g. (14.8 ml.) of Hg and 2 ml. of 1N H<sub>2</sub>SO<sub>4</sub>. The Zn dissolves completely in about 20 minutes. The liquid amalgam is repeatedly washed with very dilute H<sub>2</sub>SO<sub>4</sub>, cooled and separated in a dropping funnel from solid particles ( $\beta$  or  $\gamma$  phase, composition approx. HgZn<sub>2</sub>). These solids may be used to enhance the Zn concentration of amalgams that have lost some Zn during use. The liquid Cd and Bi amalgams (each containing about 3% of the respective metal) are prepared in an analogous fashion. However, the Bi amalgam is prepared with hydrochloric rather than sulfuric acid. The Sn amalgam (8% Sn) is prepared by heating Sn granules with Hg under hydrochloric acid. The liquid Pb amalgam (3% Pb) by heating dry Pb with the stoichiometric quantity of Hg (the starting Pb must be freed of surface oxide by treatment with conc. HCl). The amalgam product is washed with water.

PROPERTIES:

The above liquid amalgams are very stable and may be stored for a long time under weakly acidic water, with which they react extremely slowly.

Used as reducing agents in volumetric analyses.

REFERENCES:

E. Brennecke. Flüssige Amalgame als Reduktionsmittel in der Massanalyse [Liquid Amalgams as Reducing Agents in Volumetric Analysis], in: Brennecke, Fajans, Furmann, Lang and Stamm. Neuere Massanalytische Methoden [New Methods of Volumetric Analysis], Stuttgart, 1951; also contains references to original publications. C. Winterstein. Z. anal. Chem. <u>117</u>, 81 (1939).

AMALGAMS OF RARE EARTH METALS

Amalgams of rare earth metals (3% of the metal) are readily obtained by electrolysis of alcoholic solutions of the corresponding anhydrous chlorides at an Hg cathode and a graphite anode. These amalgams may be then further concentrated by distilling off the excess Hg.

REFERENCES:

E. E. Jukkola, L. F. Audrieth and B. S. Hopkins in: H. S. Booth. Inorg. Syntheses, Vol. 1, New York-London, 1939, p. 15.

## ALUMINUM AMALGAM

The normal procedure is to deposit only a surface layer of amalgam on the metal, in order to activate it for use in some specific reaction [H. Adkins, J. Amer. Chem. Soc. 44, 2175 (1922)].

## Potassium-Sodium Alloy (liquid)

The constitutional diagram of the K-Na system shows a liquidus curve minimum at -12.5 °C corresponding to 77.3 wt. % K. All alloys with compositions near this point (45-90 wt. % K) are liquids at room temperature and are much more reactive than the pure metals.

Alloys such as these are prepared by carefully heating, for instance, 3 g. of clean pieces of K and 1 g. of clean pieces of Na (or other required quantities of these metals) under anhydrous toluene or xylene, while kneading the two metals with a flat-end glass rod.

According to Lecher, the metals may be combined even at room temperature provided some ethanol is added to the protective fluid in order to activate the metal surface. The vessel is a Schlenk flask, which consists of a two-neck glass bulb (one of these necks is narrow and vertically centered, the other is inclined and somewhat to the side of the flask). An alloy useful in organic reactions may be obtained from 0.35 g. of Na and 1.6 g. of K, which are combined under weakly alcoholic ligroin while a stream of nitrogen is introduced via the side neck (the metals are kneaded by means of a flat-end glass rod introduced through the vertical neck). After a liquid alloy has been obtained, the nitrogen purge is continued while the alcoholic ligroin (including the impurities suspended in it) is replaced by clean, anhydrous ligroin.

The Schlenk flask may, of course, be replaced by other devices which allow work in the absence of air. The liquid alloy ignites spontaneously and must always be protected by an inert fluid and an inert gas  $(N_2, CO_2)$ .

**REFERENCES**:

H. Lecher. Ber. dtsch. chem. Ges. <u>48</u>, 524 (1915). For constitutional diagrams and crystal structures of the K-Na and other binary systems of alkali metals, see also A. Helms and W. Klemm. Z. anorg. allg. Chem. <u>242</u>, 201 (1939).

## Low-Melting Alloys

Low-melting alloys are often required for special purposes such as for heating baths or manometers, sealing in other liquids or gases, cementing, and for flowout devices. Some of these alloys, all of which are reasonably stable in air ar their melting points and somewhat above, are listed below.

#### 5. ALLOYS AND INTERMETALLIC COMPOUNDS

Designation	М.р., °С	Composition, wt. %
Bi-Cd-Pb-Sn eutectic (Wood's Lipowitz metal)	71	49.5 Bi/10.1 Cd/27.3 Pb/13.1 Sn
Bi-Cd-Pd eutectic	91.5	51.7 Bi/8.1 Cd/40.2 Pb
Bi-Pb-Sn eutectic	96	50 Bi/31.2 Pb/18.8 Sn
(Newton's Rose metal)		
Bi-Cd-Sn eutectic	103	54 Bi/20 Cd/26 Sn
Bi-Pb eutectic	125	56.5 Bi/43.5 Pb
Bi-Sn eutectic	139	58 Bi/42 Sn
Cd-Pb-Sn eutectic	145	18.2 Cd/32 Pb/49.8 Sn
Pb-Sn eutectic	183	37.7 Pb/62.3 Sn
Sn-Zn eutectic	198.6	91.1 Sn/8.9 Zn
Cd-Zn eutectic	266	82.6 Cd/17.4 Zn
Ag-Pb eutectic	304	2.5 Ag/97.5 Pb

Further special alloys can be obtained with Ga and In:

Designation	м.р., °С	Composition, wt. $\%$
Ga-In-Sn-Zn eutectic Ga-In-Sn eutectic Ga-In-Zn eutectic Ga-In eutectic Ga-Sn eutectic Ga-Zn eutectic L 46	3 5 13 16 20 25 46.5	61 Ga/25 In/13 Sn/1 Zn 62 Ga/25 In/13 Sn 67 Ga/29 In/4 Zn 76 Ga/24 In 92 Ga/8 Sn 95 Ga/5 Zn 40.6 Bi/8.2 Cd/18 In/22.4 Pb/ 10.8 Sn
L 58 In-Sn eutectic	58 117	49 Bi/21 In/18 Pb/12 Sn 52 In/48 Sn

An exhaustive review of alloys melting between -39 °C and +419 °C is given by Spengler. If the low-melting alloys are used as cements or solders for nonmetallic objects, or if they are melted in nonmetallic vessels, then the thermal expansion coefficients must be carefully considered. For instance, Wood's metal may, on cooling and resolidification, burst glass vessels in which it is contained for use as a bath liquid.

REFERENCES:

H. Spengler. Metall <u>9</u>, 682 (1955); Z. Metallkunde <u>46</u>, 464 (1955);
 J. D'Ans and E. Lax. Taschenbuch f. Chemiker and Physiker

[Handbook for Chemists and Physicists], Berlin-Göttingen-Heidelberg, 1959; M. Hansen and K. Anderko. Constitution of Binary Alloys, New York-Toronto-London, 1958; C. J. S. Smithells. Metals Reference Book, London, 1949; M. T. Ludwich (The Indium Corp. of Amer.), Indium etc., New York, 1950; W. Kroll. Metallwirtschaft <u>11</u>, 435 (1932).

## Formula Index

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2

NOTE: Rare earth elements (lanthanides) are designated by the common symbol Ln (except in special cases where R.E. has been used). Thus, for their compounds see under Ln.

A

Ag 1029, 279, 1028 Ag<sub>2</sub>C<sub>2</sub> 1047 Ag(CF<sub>3</sub>COO) 205  $Ag_2C_4H_4O_6$  1049 AgCN 661 Ag<sub>2</sub>CN<sub>2</sub> 1047 Ag<sub>2</sub>CO<sub>3</sub> 1048 AgClO<sub>3</sub> 1037  $[Ag(dipyr)_3](ClO_4)_2$ 105**0** [Ag(dipyr)2]NO3 1050 [Ag(dipyr)<sub>3</sub>](NO<sub>3</sub>)<sub>2</sub> 1051 [Ag(dipyr)<sub>2</sub>]S<sub>2</sub>O<sub>8</sub> 1051 AgF 240 AgF<sub>2</sub> 241 Ag<sub>2</sub>F 239 AgI 1035 AgMnO<sub>4</sub> 1463 AgN<sub>3</sub> 1045 Ag<sub>3</sub>N 1046 AgNCS 671 AgNH<sub>2</sub> 1043 AgNO<sub>2</sub> 1048 Ag<sub>2</sub>N<sub>2</sub>O<sub>2</sub> 493, 514 Ag<sub>2</sub>O 1037 Ag<sub>2</sub>O<sub>2</sub> 1038 Ag<sub>2</sub>O·4WO<sub>3</sub>·aq 1728  $AgPO_2(NH_2)_2$  582  $[Agphen_2]S_2O_8$  1050 Ag<sub>2</sub>S 1039 Ag<sub>2</sub>SO<sub>3</sub> 1043 Ag<sub>2</sub>SO<sub>4</sub> 1042

Ag<sub>2</sub>Se 1041 Ag<sub>2</sub>SiO<sub>3</sub> 705 Ag<sub>2</sub>Te 1042 AlAs 831 AlAsO4 831 AlB<sub>2</sub> 772, 1792 AlB<sub>12</sub> 772 AlBr<sub>3</sub> 806, 813 AlBr<sub>3</sub>·H<sub>2</sub>S 819 Al<sub>4</sub>C<sub>3</sub> 832 Al(C<sub>2</sub>H<sub>5</sub>)<sub>3</sub> 810  $Al(C_2H_5)_2Br 809$ Al(CH<sub>3</sub>COO)<sub>3</sub> 835  $Al(C_2H_5)_2Cl \cdot O(C_2H_5)_2$ 811 Al(C<sub>2</sub>H<sub>5</sub>)<sub>2</sub>H 811 Al(C<sub>5</sub>H<sub>7</sub>O<sub>2</sub>)<sub>3</sub> 836  $Al(C_2H_5)_3 \cdot O(C_2H_5)_2$ 811 Al(CN)<sub>3</sub>.O(C<sub>2</sub>H<sub>5</sub>)<sub>2</sub> 834 Al<sub>4</sub>Ce 1792 AICl<sub>3</sub> 680, 805, 812 Al<sub>2</sub>Cl<sub>3</sub>H<sub>3</sub> 808 AlCl<sub>3</sub>.6H<sub>2</sub>O 815 AlCl<sub>3</sub>.NH<sub>3</sub> 817 AlCl<sub>3</sub>·PCl<sub>5</sub> 818 AlCl<sub>3</sub>.SO<sub>2</sub> 817 Al<sub>2</sub>Cl<sub>6</sub>·SOCl<sub>2</sub> 818 AlF<sub>3</sub> 225 AlF<sub>3</sub>.3H<sub>2</sub>O 225  $AlH_3 \cdot N(CH_3)_3 809$ AlH<sub>3</sub>·2N(CH<sub>3</sub>)<sub>3</sub> 809  $(AlH_3)_n \cdot xO(C_2H_5)_2 807$ AlI<sub>3</sub> 814

All<sub>3</sub>.6NH<sub>3</sub> 819 Al<sub>4</sub>La 1792 AlN 827 Al(N<sub>3</sub>)<sub>3</sub> 829 Al<sub>3</sub>Nb 1792 Al<sub>2</sub>O<sub>3</sub> 824, 1660 Al(OCH<sub>3</sub>)<sub>3</sub> 833 Al(OC<sub>2</sub>H<sub>5</sub>)<sub>3</sub> 834, 835 Al(OC<sub>2</sub>H<sub>4</sub>)<sub>3</sub>N 835 AI(OD)<sub>3</sub> 134 Al(OH)<sub>3</sub> 676, 810, 820 Alooh 820 Al<sub>2</sub>O<sub>3</sub>·2SO<sub>2</sub>·H<sub>2</sub>O 824 Al<sub>2</sub>O<sub>3</sub>•3SO<sub>2</sub>•xH<sub>2</sub>O 824 AlP 829 AlPO<sub>4</sub> 831 Al<sub>2</sub>S<sub>3</sub> 134, 700, 823 AISb 831 Al<sub>2</sub>Se<sub>3</sub> 825 Al<sub>3</sub>Ta 1792 Al<sub>2</sub>Te<sub>3</sub> 826 Al<sub>3</sub>Th 1792 Al<sub>3</sub>V 1792 Al<sub>3</sub>Zr 1792 Ar 82 As 591 AsBr3 597 AsCl<sub>3</sub> 596 AsF<sub>3</sub> 179, 197 AsF<sub>5</sub> 198 AsH<sub>3</sub> 593 As<sub>2</sub>H<sub>4</sub> 594 AsI<sub>2</sub> 598 AsI<sub>3</sub> 597

$As_2O_3$ 600	B(OCH <sub>3</sub> ) <sub>3</sub> 797	Ba
As <sub>2</sub> O <sub>5</sub> 601	B <sub>3</sub> O <sub>3</sub> (CH <sub>3</sub> ) <sub>3</sub> 800	Ba
$A_{s_2}O_{5} + \frac{3}{3}H_2O_{601}$	$B_{3}O_{3}(n-C_{4}H_{9})_{3}$ 801	Ba
As205.6MoO3.ag 1736	B(OH) <sub>2</sub> CH <sub>3</sub> 800	Ba
As 205. 18 Mo 0 3. ag	$B(OH)_2(n-C_4H_9) = 801$	Ba
1735	B <sub>2</sub> O <sub>3</sub> ,24WO <sub>3</sub> ,ag 1716	Ba
As 2S5 603	BPO <sub>4</sub> 796	Ba
As <sub>4</sub> S <sub>4</sub> 603	$B_2S_3$ 788	Ba
$As_2Zn_3$ 594	Ba 922	Ba
Au 1052	BaBr <sub>2</sub> 930	Ba
Au <sub>2</sub> C <sub>2</sub> 1063	$Ba(BrO_3)_2 \cdot H_2O_3 = 316$	Ba
AuCN 1064	BaCO <sub>3</sub> 933	Be
AuCl 1055	$BaCS_3$ 674	Be
AuCl <sub>3</sub> 1056	$B_aCl_2$ 930	Be
$Au_2O_3 = 1059$	$B_{a}(ClO_{3})_{2} \cdot H_{2}O_{3}14$	Be
$Au(OH)_3 = 1060$	$Ba(ClO_4)_2$ 320	Be
AuS 1062	$B_{a}(ClO_{4})_{2} \cdot 3H_{2}O_{320}$	Be
Au <sub>2</sub> S 1061	Ba <sub>2</sub> CrO <sub>4</sub> 1393	Be
Au <sub>2</sub> S <sub>3</sub> 1063	$Ba_3(CrO_4)_2$ 1394	Be
200 1000	$Ba_2[Cu(OH)_6]$ 1685	Be
	$BaF_2$ 234	Be
D.	Ba <sub>3</sub> [Fe(OH) <sub>6</sub> ] <sub>2</sub> 1691	Be
В	$Ba_{2}[Fe(OH)_{7}] \cdot \frac{1}{2}H_{2}O$ 1691	Be
	BaGeF <sub>6</sub> 215	Be
B 770	BaH <sub>2</sub> 929	Be
B <sub>2</sub> Al 772	$Ba_{3}H_{4}(IO_{6})_{2}$ 326	Be
B <sub>12</sub> Al 772	$Ba(H_2PO_2)_2 \cdot H_2O_557$	Be
BAsO4 797	$BaH_2P_2O_6 \cdot 2H_2O_562$	Be
BBr <sub>3</sub> 770, 781	$BaI_2$ 930	Be
B(CH <sub>3</sub> ) <sub>3</sub> 798	$B_{a}(M_{n}O_{4})_{2}$ 1462	Be
$B(C_2H_5)_3$ 799	$Ba(N_3)_2 942$	Bi
$B(C_nH_{2n+1})_3 800$	$Ba(N_3)_2 \cdot H_2O 942$	Bi
BCl <sub>3</sub> 780	Ba <sub>3</sub> N <sub>2</sub> 940	Bi
$BCl_2(C_nH_{2n+1})$ 803	BaO 933	Bi
BF <sub>3</sub> 219	BaO <sub>2</sub> 937	Bi
$BF_{2}(n-C_{4}H_{9})$ 802	$B_{a}O_{2}.8H_{2}O_{936}$	Bi
BF <sub>3</sub> .2H <sub>2</sub> O 784	3BaO.P2O5.24M0O3.ag	Bi
BF <sub>3</sub> .NH <sub>3</sub> 785	1732	Bi
$BF_{3} \cdot O(C_{2}H_{5})_{2}$ 786	10BaO·P <sub>2</sub> O <sub>5</sub> ·24V <sub>2</sub> O <sub>5</sub> ·aq	Bi
B <sub>2</sub> H <sub>6</sub> 773	1740	Bi
BH3·N(CH3)3 778	3BaO·P2O5·24WO3·ag	Bi
BI <sub>3</sub> 782	1721	Bi
BN 789	Ba <sub>3</sub> (PO <sub>2</sub> S <sub>2</sub> ) <sub>2</sub> .8H <sub>2</sub> O 572	Bi
B(N <sub>3</sub> ) <sub>3</sub> 476	BaPt(CN)4.4H20 1576	Bi
B <sub>3</sub> N <sub>3</sub> Cl <sub>3</sub> H <sub>3</sub> 779	BaReO <sub>4</sub> 1485	Bi
B <sub>3</sub> N <sub>3</sub> H <sub>6</sub> 779	$Ba(ReO_4)_2$ 1485	Bi
<b>D</b> • • • •		

aS 938  $(SO_3F)_2$  173  $aS_2O_6 \cdot 2H_2O_397$ aSO<sub>4</sub>-KMnO<sub>4</sub> 1463 aSe 939 aSeO₄ 939 aSi 947 aSiO<sub>3</sub> 706 aSi<sub>2</sub>O<sub>5</sub> 706  $a[Sn_2O(OH)_4]$  1696 aTe 940 e 887 eBr<sub>2</sub> 891 eC<sub>2</sub> 899, 900 e<sub>2</sub>C 899 e(CH<sub>3</sub>COO)<sub>2</sub> 901 eCO<sub>3</sub> 893 eCl<sub>2</sub> 889 eF<sub>2</sub> 231 eI2 892 e(N<sub>3</sub>)<sub>2</sub> 899 e<sub>3</sub>N<sub>2</sub> 898 eO 893  $e_4O(CH_3COO)_6$  901  $e_4O(C_2H_5COO)_6$  902 e(OH)<sub>2</sub> 894  $e_4O(HCOO)_6$  902 eS 895 eSe 897 eTe 897 620 BO3.2H2O 627 Br<sub>3</sub> 623 Cl<sub>2</sub> 622 Cl<sub>3</sub> 621 F<sub>3</sub> 201 F<sub>5</sub> 202 I<sub>3</sub> 624 ICl<sub>2</sub> 622 <sub>2</sub>O<sub>3</sub> 620  $_{2}O_{4} \cdot aq 629$ **OBr** 624 OC1 622 OI 625 ONO<sub>2</sub> 626 ONO2.1/H2O 626 BiONO<sub>3</sub> 620

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2

## С

C 630 C30A1Cl4.2A1Cl3 644 C<sub>8</sub>Br 643 CClF<sub>3</sub> 205 CCl<sub>2</sub>F<sub>2</sub> 151, 205  $C_2Cl_2F_4$  205 C<sub>2</sub>Cl<sub>3</sub>F<sub>3</sub> 205 C<sub>8</sub>Cs 635 C<sub>24</sub>Cs 635 C<sub>36</sub>Cs 635 C<sub>48</sub>Cs 635 C<sub>12</sub>Cs(NH<sub>3</sub>)<sub>2</sub> 637 CF 640 CF<sub>4</sub> 203, 207 C<sub>4</sub>F 641 C<sub>2</sub>HCl<sub>3</sub>F<sub>2</sub> 205 CHF<sub>3</sub> 204  $C_{24}HSO_{4} \cdot 2H_{2}SO_{4}$  642 CIF<sub>3</sub> 205 C<sub>8</sub>K 635 C<sub>24</sub>K 635 C<sub>36</sub>K 636 C<sub>48</sub>K 636  $C_{12}K(NH_3)_2$  637  $C_{12}Li(NH_3)_2$  637 (CN)<sub>2</sub> 661 CNBr 665 CNC1 662

CNI 666  $C_{12}Na(NH_3)_2$  637 CO 645 CO<sub>2</sub> 647  $C_{3}O_{2}$  648 COBrF 210 COCl<sub>2</sub> 650 COC1F 208 COF<sub>2</sub> 206, 210 COIF 211  $(CONH)_3$  668 COS 654 COSe 655 C<sub>8</sub>Rb 635 C24Rb 635 C<sub>36</sub>Rb 635 C<sub>48</sub>Rb 635  $C_{12}Rb(NH_3)_2$  637 CS<sub>2</sub> 652 C<sub>3</sub>S<sub>2</sub> 653  $C_{3}S_{2}Br_{6}$  653 CSe<sub>2</sub> 656 Ca 922 Ca(A1H<sub>4</sub>)<sub>2</sub> 806 CaBr<sub>2</sub> 930 CaC<sub>2</sub> 943 CaCN<sub>2</sub> 946 CaCO<sub>3</sub> 931 CaCl<sub>2</sub> 930 Ca(ClO<sub>4</sub>)<sub>2</sub> 320  $C_{a}(ClO_{4})_{2} \cdot 4H_{2}O_{320}$ CaF<sub>2</sub> 233 CaGe 948 CaH<sub>2</sub> 929 CaI<sub>2</sub> 930 Ca<sub>3</sub>N<sub>2</sub> 940 CaO 931 CaO<sub>2</sub> 936 CaO2.8H2O 936 Ca(OH)<sub>2</sub> 934  $Ca_{3}P_{2}$  942 Ca<sub>10</sub>(PO<sub>4</sub>)<sub>6</sub>(OH)<sub>2</sub> 545  $Ca_2PbO_4$  760 CaS 938 CaSe 939 CaSeO<sub>4</sub> 939

CaSi 946 CaSi<sub>2</sub> 946 CaTe 940 Cd 1042 CdAs<sub>2</sub> 1103 Cd<sub>3</sub>As<sub>2</sub> 1103 CdBr<sub>2</sub> 1096 Cd(C<sub>2</sub>H<sub>5</sub>)<sub>2</sub> 1103 Cd(CH<sub>3</sub>COO)<sub>2</sub> 1105 Cd(CN)<sub>2</sub> 1105 CdCO<sub>3</sub> 1104 CdCl<sub>2</sub> 1093 CdCl<sub>2</sub>.KCl·H<sub>2</sub>O 1095 CdF<sub>2</sub> 243 CdFe<sub>2</sub>O<sub>4</sub> 1107 CdI<sub>2</sub> 1096 Cd<sub>3</sub>N<sub>2</sub> 1100 Cd(NH<sub>2</sub>)<sub>2</sub> 1100 Cd(OH)<sub>2</sub> 1097 Cd(OH)Cl 1094 CdP<sub>2</sub> 1101 CdP<sub>4</sub> 1101 Cd<sub>3</sub>P<sub>2</sub> 1101 CdS 1098 Cd(SCN)<sub>2</sub> 1106 CdSe 1099 Cd<sub>2</sub>SiO<sub>4</sub> 1107 CeF<sub>3</sub> 247 CeF<sub>4</sub> 247 CeO<sub>2</sub> 1132 Ce<sub>2</sub>O<sub>3</sub> 1151 CeS 1155 Cl<sub>2</sub> 272 (CIBNH)<sub>3</sub> 779 CICN 662 CIF 153 ClF<sub>3</sub> 155 Cl<sub>2</sub>·6H<sub>2</sub>O 274 ClN<sub>3</sub> 476 CINO<sub>3</sub> 326  $CIN(SO_3K)_2$  508 ClO<sub>2</sub> 301 Cl<sub>2</sub>O 299  $Cl_2O_6$  303 Cl<sub>2</sub>O<sub>7</sub> 304 ClO<sub>2</sub>F 165

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 ClO<sub>3</sub>F 166 ClO4F 167 [ClPy<sub>x</sub>]NO<sub>3</sub> 328 Co 1513 CoAl2O4 1525 CoBr<sub>2</sub> 1517 CoBr<sub>2</sub>.6H<sub>2</sub>O 1517 Co<sub>2</sub>C 1531 [Co(CO)<sub>3</sub>]<sub>4</sub> 1746  $[C_0(CO)_4]_2$  1746 Co(CO)<sub>4</sub>H 1753 CoCl<sub>2</sub> 267, 1515 [Co en 3]Br 3 1538  $[Co en_3]Br_3 \cdot 2H_2O$ 1539 [Co en<sub>3</sub>]Br<sub>3</sub>·3H<sub>2</sub>O 1538 CoF<sub>2</sub> 267 CoF<sub>3</sub> 268 CoI<sub>2</sub> 1518 CoI2.6H2O 1518 CoN 1529 Co<sub>2</sub>N 1529 Co(NH<sub>2</sub>)<sub>3</sub> 1526  $[C_0(NH_3)_4CO_3]_2SO_4$ •3H<sub>2</sub>O 1535 [Co(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>2</sub> 1516 [Co(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>3</sub> 1531  $[C_0(NH_3)_5Cl]Cl_2$ 1532 $[C_0(NH_3)_4CI]Cl_2$  1536  $[C_0(NH_3)_4Cl_2]Cl_1$ ·½H₂O 1536  $[Co(NH_3)_4Cl_2]Cl_1$ •H<sub>2</sub>O 1537  $[C_0(NH_3)_5(H_2O)]_2$ (C<sub>2</sub>O<sub>4</sub>)<sub>3</sub>.4H<sub>2</sub>O 1532  $[C_0(NH_3)_4(H_2O)_2]_2$ (SO<sub>4</sub>)<sub>3</sub>.3H<sub>2</sub>O 1537 [Co(NH<sub>3</sub>)<sub>6</sub>](NO<sub>3</sub>)<sub>3</sub> 1526  $[C_0(NH_3)_5NO_2]Cl_2$ 1534  $[Co^{III}(NH_3)_5(O_2)Co^{IV}]$  $(NH_3)_5](SO_4)_2 \cdot SO_4H \cdot$ 3H<sub>2</sub>O 1540

[Co(NH<sub>3</sub>)<sub>5</sub>ONO]Cl<sub>2</sub> 1535 Co(NO)2Br 1761 Co(NO)(CO)3 1761 Co(NO)2Cl 1761 Co(NO)<sub>2</sub>I 1761 [Co(NO<sub>2</sub>)<sub>6</sub>]Na<sub>3</sub> 1541  $[C_0(NO)_2(S_2O_3)_2]K_3$ 1766 CoO 1514, 1519 Co<sub>3</sub>O<sub>4</sub> 1520 Co(OH)<sub>2</sub> 1521 CoO(OH) 1520 CoP 1530 CoP<sub>3</sub> 1530 Co<sub>2</sub>P 1530 CoS 1523 CoS<sub>2</sub> 1523 Co<sub>3</sub>S<sub>4</sub> 1523 Co<sub>9</sub>S<sub>8</sub> 1523 CoSO2+3H2O 393 Co<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub>.18H<sub>2</sub>O 1524 Cr 1334 CrBr<sub>2</sub> 1340 CrBr<sub>3</sub> 1341  $Cr(C_6H_6)_2$  1397 Cr(C<sub>12</sub>H<sub>10</sub>)<sub>2</sub> 1397  $[Cr(C_{12}H_{10})(C_{6}H_{6})]I$ 1398  $Cr_2(CH_3COO)_4 \cdot 2H_2O$ 1368[Cr(C<sub>6</sub>H<sub>6</sub>)<sub>2</sub>]I 1395  $[Cr(C_{12}H_{10})_2]I$  1396  $[Cr(C_6H_5NC)_6]$  1363 Cr(C<sub>5</sub>H<sub>7</sub>O<sub>2</sub>)<sub>3</sub> 1383 Cr(C<sub>2</sub>H<sub>5</sub>OCS<sub>2</sub>)<sub>3</sub> 1383 Cr(CO)<sub>6</sub> 1741 CrC<sub>2</sub>O<sub>4</sub>·2H<sub>2</sub>O 1370 Cr(CO)<sub>3</sub>Py<sub>3</sub> 1749 Cr(CO)<sub>4</sub>Py<sub>2</sub> 1749 CrCl<sub>2</sub> 1336 CrCl<sub>3</sub> 1338  $[CrCl_{3}(C_{2}H_{5}OH)_{3}]$  1380 [CrCl en 2]Cl 1357  $[CrCl_2 en_2]Cl H_2O 1356$ [CrCl<sub>3</sub>(NH<sub>3</sub>)<sub>3</sub>] 1381  $[CrCl(NH_3)_5]Cl_2$  1352

 $[CrCl_3(OH_2)_3]$  1380  $[CrCl(OH_2)_5]Cl_2 H_2O$ 1350  $[CrCl_2(OH_2)(NH_3)_3]Cl$ 1358 [CrCl<sub>3</sub>Py<sub>3</sub>] 1381 [Cr(Dipy)<sub>3</sub>] 1363 [Cr(Dipy)<sub>3</sub>]ClO<sub>4</sub> 1362  $[Cr(Dipy)_3](ClO_4)_2$ 1361 [Cr en3]C13.3.5H2O 1354  $[Cr en_3](SCN)_3 \cdot H_2O$ 1354 $[Cr en_3]_2(SO_4)_3$  1354 CrF<sub>2</sub> 256 CrF<sub>3</sub> 258 CrF.3H<sub>2</sub>O 258 CrF<sub>4</sub> 258 CrI<sub>2</sub> 1341 CrI<sub>3</sub> 1344 CrN 1347 Cr(NH<sub>2</sub>CH<sub>2</sub>COO)<sub>3</sub> 1382 [Cr(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>3</sub> 1351  $[Cr(NH_3)_6](NO_3)_3$  1351  $[Cr(NH_3)_5(OH)Cr(NH_3)_5]$ Cl<sub>5</sub> 1359  $[Cr(NH_3)_5(OH)Cr(NH_3)_4]$ (OH<sub>2</sub>)]Cl<sub>5</sub> 1360  $[Cr(OCN_2H_4)_6]Cl_3\cdot 3H_2O$ 1359 CrO<sub>2</sub>Cl<sub>2</sub> 1384 CrO<sub>2</sub>(ClO<sub>4</sub>)<sub>2</sub> 1387 CrO<sub>2</sub>F<sub>2</sub> 258  $Cr(OH)_{3} \cdot nH_2O$  1345  $[Cr(OH_2)_6](CH_3COO)_3$ 1371 $[Cr_3(OH)_2(CH_3COO)_6]$ (CH<sub>3</sub>COO) • nH<sub>2</sub>O 1371  $[Cr_3(OH)_2(CH_3COO)_6]$ Cl.8H<sub>2</sub>O 1371  $[Cr(OH_2)_6]Cl_3$  1348 [Cr(OH)<sub>6</sub>]Na<sub>3</sub> 1688 CrO<sub>2</sub>(NO<sub>3</sub>)<sub>2</sub> 1386 CrO<sub>3</sub>·2Py 1385 CrPO<sub>4</sub> 1364

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 CrS 1346 Cr<sub>2</sub>S<sub>3</sub> 1346  $[Cr(SCN)_4(C_6H_5NH_2)_2]$ NH<sub>4</sub>.1½H<sub>2</sub>O 1378 [Cr(SCN)<sub>2</sub>en<sub>2</sub>]SCN 1357 CrSO<sub>4</sub>.5H<sub>2</sub>O 1365 Cs 958  $C_{sAl}(SO_{4})_{2} \cdot 12H_{2}O_{956}$ CsBrCl<sub>2</sub> 294 CsC<sub>8</sub> 635 CsC<sub>24</sub> 635 CsC<sub>36</sub> 636 CsC<sub>48</sub> 636 Cs<sub>2</sub>CO<sub>3</sub> 987 CsCl 951, 955  $Cs_2CrO_4$  1389 Cs<sub>2</sub>Cr<sub>2</sub>O<sub>7</sub> 1389 CsGe 989 CsH 971 CsIBr<sub>2</sub> 297 CsICl<sub>2</sub> 296  $C_{s}Mn(SO_{4})_{2} \cdot 12H_{2}O$  1468 CsN<sub>3</sub> 476  $C_{s}(NH_{3})_{2}C_{12}$  637 CsO<sub>2</sub> 981 Cs<sub>2</sub>O 974 CsOH 983  $Cs_2S_2$  369  $Cs_2S_3$  369  $C_{s_2}S_5$  369  $Cs_2S_6$  369 Cs<sub>2</sub>SeCl<sub>6</sub> 425 CsSi 989 2Cs20.SiO2.12MoO3.aq 1730 Cs<sub>2</sub>TeCl<sub>6</sub> 444  $C_{s_3}(Tl_2Cl_9)$  874 Cu 1003, 1633 CuBr 1006 CuBr<sub>2</sub> 1009 Cu<sub>2</sub>C<sub>2</sub>·H<sub>2</sub>O 1026  $CuCO_3 \cdot Cu(OH)_2$  1024 2CuCO<sub>3</sub>·Cu(OH)<sub>2</sub> 1025 CuCl 1005 CuCl<sub>2</sub> 1008 CuCl<sub>2</sub>·Cu(OH)<sub>2</sub> 1010 EuI<sub>2</sub> 1150

CuF<sub>2</sub> 238 CuF2.5H2O.5HF 238 CuH 1004 Cu<sub>2</sub>HgI<sub>4</sub> 1110 CuI 1007  $Cu(N_3)_2$  1022 Cu<sub>3</sub>N 1021  $[Cu(NH_3)_4]SO_4 \cdot H_2O$ 1021 CuO 1012 Cu<sub>2</sub>O 1011 Cu(OH)<sub>2</sub> 1013 [Cu(OH)<sub>4</sub>]Na<sub>2</sub> 1684 CuP<sub>2</sub> 1024 Cu<sub>3</sub>P 1023 Cu<sub>2</sub>P<sub>4</sub>O<sub>12</sub> 553 CuS 1017 Cu<sub>2</sub>S 1016 Cu<sub>2</sub>SO<sub>4</sub> 1020 Cu<sub>2</sub>Se 1019 Cu<sub>2</sub>Te 1019 D D<sub>2</sub> 121 DBr 131 DC1 129 DF 127 DH 126 DI 133 D<sub>2</sub>O 119 D<sub>3</sub>PO<sub>3</sub> 132 D<sub>3</sub>PO<sub>4</sub> 138 D<sub>2</sub>S 134 D<sub>2</sub>SO<sub>4</sub> 135 Е EuBr<sub>2</sub> 1149 EuCO<sub>3</sub> 1137 EuCl<sub>2</sub> 1150 EuCl<sub>2</sub>·2H<sub>2</sub>O 1136  $EuF_2$  248

EuO 1156 EuS 1155 EuSO<sub>4</sub> 1137 EuSe 1155 EuTe 1155 F F<sub>2</sub> 143 F<sub>2</sub>O 163  $F_{2}O_{2}$  162 FSO<sub>2</sub>NO 186 Fe 1490 FeBr<sub>2</sub> 1493 FeBr<sub>3</sub> 1494 Fe<sub>3</sub>C 1503,1792  $[Fe_3(CH_3COO)_6(OH)_2]$ CH<sub>3</sub>COO·H<sub>2</sub>O 1508 [Fe(CN)<sub>5</sub>(CO)]Na<sub>3</sub> 1769  $[Fe(CN)_5NH_3]Na_2 H_2O$ 1512 $[Fe(CN)_5NH_3]Na_3\cdot 3H_2O$ 1511  $[Fe(CN)_5(NO)]Na_2 \cdot 2H_2O$ 1768 [Fe(CN)5(OH)2]Na3 1769 [Fe(CO)<sub>4</sub>]<sub>3</sub> 1745 Fe(CO)<sub>5</sub> 1743 Fe<sub>2</sub>(CO)<sub>9</sub> 1744 Fe(CO)<sub>4</sub>Br<sub>2</sub> 1751 Fe(CO)<sub>4</sub>Cl<sub>2</sub> 1751 Fe(CO)<sub>4</sub>H<sub>2</sub> 1752 Fe(CO)<sub>4</sub>Hg 1755 Fe(CO)<sub>4</sub>I<sub>2</sub> 1751 [Fe(CO)<sub>3</sub>NO]K 1759 FeCl<sub>2</sub> 1491 FeCl<sub>3</sub> 1492 [Fe en3][Fe2(CO)8] 1756 FeF<sub>2</sub> 266 FeF<sub>3</sub> 266 Fel<sub>2</sub> 1495 Fe<sub>2</sub>N 1502 Fe<sub>4</sub>N 1502

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2  $Fe(NO)_2(CO)_2$  1760  $[Fe(NO)_2SC_2H_5]_2$  1765  $[Fe(NO)_2S]Na\cdot 4H_2O$ 1763  $[Fe(NO)_2S_2O_3]K\cdot H_2O$ 1766  $[Fe(NO)_2S_2O_3]Na 1766$ FeO 1497 Fe<sub>2</sub>O<sub>3</sub> 1661 Fe<sub>3</sub>O<sub>4</sub> 1499 FeOC1 1501 Fe(OH)<sub>2</sub> 1498 [Fe(OH)<sub>4</sub>]Na<sub>2</sub> 1686  $[Fe(OH)_7]Na_4 \cdot 2H_2O$ 1689  $[Fe(OH)_8]Na_5 \cdot 5H_2O$ 1690 FeO(OH) 1499  $3Fe_2O_3 \cdot 4SO_3 \cdot 9H_2O$ 1507  $[FePy_6][Fe_4(CO)_{13}]$ 1758FeS 1502  $[Fe(SCN)_6]Na_3 \cdot 12H_2O$ 1511  $Fe_3(SO_4)_2(OH)_5 \cdot 2H_2O$ 1507

## G

Ga 837 GaAs 857 GaBr<sub>2</sub> 846 GaBr<sub>3</sub> 845 Ga(CH<sub>3</sub>)<sub>3</sub> 840 Ga(CH<sub>3</sub>)<sub>3</sub> $\cdot$ N(C<sub>2</sub>H<sub>5</sub>)<sub>3</sub> 841 GaCl<sub>2</sub> 846 GaCl<sub>3</sub> 843 Ga(ClO<sub>4</sub>)<sub>3</sub> $\cdot$ GH<sub>2</sub>O 839 GaF<sub>3</sub> 227 Ga<sub>2</sub>H<sub>6</sub> 840 Ga<sub>2</sub>H<sub>2</sub>(CH<sub>3</sub>)<sub>4</sub> 840 GaI<sub>3</sub> 846 GaN 855 Ga(N<sub>3</sub>)<sub>3</sub> 476 Ga(NO<sub>3</sub>)<sub>3</sub> 856 Ga<sub>2</sub>O 849  $Ga_2O_3 848$ Ga(OH)<sub>3</sub> 847 GaO(OH) 847 GaP 857 GaS 851 Ga<sub>2</sub>S 852 Ga<sub>2</sub>S<sub>3</sub> 850 GaSb 857 GaSe 854 Ga<sub>2</sub>Se 854 Ga<sub>2</sub>Se<sub>3</sub> 854 GaTe 855 Ga<sub>2</sub>Te<sub>3</sub> 855 Ge 712 GeBr<sub>4</sub> 718 Ge(CH<sub>3</sub>COO)<sub>4</sub> 726 GeCH<sub>3</sub>I<sub>3</sub> 722 GeCl<sub>2</sub> 716 GeCl<sub>4</sub> 707, 715 GeF<sub>4</sub> 215 GeH<sub>4</sub> 713 Ge<sub>2</sub>H<sub>6</sub> 713 Ge<sub>3</sub>H<sub>8</sub> 713 GeHCl<sub>3</sub> 717,721 GeI<sub>2</sub> 720 GeI<sub>4</sub> 719 Ge<sub>3</sub>N<sub>4</sub> 722 Ge(NH)<sub>2</sub> 723 Ge<sub>2</sub>N<sub>3</sub>H 723 GeO 711 GeO<sub>2</sub> 706  $Ge(OC_2H_5)_4$  725 GeS 723, 724 GeS<sub>2</sub> 723

## H

 $\begin{array}{l} H_2 \ 111 \\ HA1Br_4 \cdot 2O(C_2H_5)_2 \ 817 \\ HA1C1_4 \cdot 2O(C_2H_5)_2 \ 816 \\ H_3AsO_4 \ 601 \\ H_3AsO_4 \cdot \frac{1}{2}H_2O \ 601 \\ H_7AsO_6 \ 601 \end{array}$ 

HAuCl<sub>4</sub>.4H<sub>2</sub>O 1057 HBF<sub>4</sub> 221 H[BF<sub>2</sub>(OH)<sub>2</sub>] 784 HBO<sub>2</sub> 791 HBr 282 HBrO<sub>3</sub> 315 HCN 658, 668  $H_2CS_3$  674 HC1 280 HC10 308 HClO<sub>3</sub> 312 HClO<sub>4</sub> 318  $H_3C_0(CN)_6$  1542 H<sub>3</sub>Co(CN)<sub>6</sub>.5H<sub>2</sub>O 1543  $H[Cr(SCN)_4(NH_3)_2]$ 1377 HD 126 HF 145 H<sub>3</sub>Fe(CN)<sub>6</sub> 1510 H<sub>4</sub>Fe(CN)<sub>6</sub> 1509 H<sub>2</sub>IrCl<sub>6</sub> 1593 HI 286 HICl<sub>4</sub>.4H<sub>2</sub>O 299 HIO<sub>3</sub> 316  $H_5IO_6$  322  $HIO_3 \cdot I_2O_5$  307  $H_2M_0O_4 \cdot H_2O_1412$ HN<sub>3</sub> 472 HNCO 667, 668 HNCS 669 HNO<sub>3</sub> 491  $H_2N_2O_2$  492  $HNbO_4 \cdot nH_2O$  1324 H<sub>2</sub>O 117 H<sub>2</sub>O<sub>2</sub> 140 H<sub>3</sub>PO<sub>2</sub> 555 H<sub>3</sub>PO<sub>3</sub> 554 H<sub>3</sub>PO<sub>4</sub> 543 H<sub>4</sub>P<sub>2</sub>O<sub>6</sub> 558 H<sub>4</sub>P<sub>2</sub>O<sub>6</sub>·2H<sub>2</sub>O 559 H<sub>4</sub>P<sub>2</sub>O<sub>7</sub> 546 HPO<sub>2</sub>Cl<sub>2</sub> 538 HPO<sub>2</sub>(NH<sub>2</sub>)<sub>2</sub> 582  $H_2PO_3NH_2$  579 H<sub>3</sub>PO<sub>3</sub>S 568 H<sub>2</sub>PtCl<sub>4</sub> 1570

pp 1-992: Vol. 1; pp 1003-1	810: Vol. 2	
H <sub>2</sub> PtCl <sub>6</sub> .6H <sub>2</sub> O 1569	Hg 28	In 857
$H_{2}S_{344}$	HgBr <sub>2</sub> 1109	InAs 867
$H_{2}S_{2}$ 350	$H_{g}(C_{2}H_{5})_{2}$ 1118	InBr 862
$H_{2}S_{3}$ 350	$H_{g}(CH_{3}COO)_{2}$ 1120	InBr <sub>2</sub> 861
$H_{2}S_{4}$ 353	Hg <sub>2</sub> (CH <sub>3</sub> COO) <sub>2</sub> 1120	InBr <sub>3</sub> 859
$H_{2}S_{5} = 353$	$H_{g}(CN)_{2}$ 1121	InCl 862
$H_{2}S_{6}$ 353, 355	HgCO <sub>3</sub> 243	InCl <sub>2</sub> 861
$H_{2}S_{7}$ 353, 355	HgCl <sub>2</sub> , 4HgO 1108	InCl <sub>3</sub> 858
$H_2S_8$ 353, 355	HgCl <sub>2</sub> ·2NH <sub>3</sub> 1114	InF <sub>3</sub> 228
$H_{2}S_{*}$ 346	$HgF_2$ 244	InI 862
H <sub>2</sub> SO <sub>5</sub> 388	$H_{g_2}F_2 = 243$	InI <sub>2</sub> 861
$H_{2}S_{2}O_{8}$ 389	$Hg_2(NH)Br_2$ 1115	InI <sub>3</sub> 860
$H_2S_xO_3$ 405	HgNH <sub>2</sub> Cl 1114	InN 866
$H_2S_{x}O_6$ 405	$[Hg(NS)_2]_{x}$ 1118	In <sub>2</sub> O 863
HSO <sub>3</sub> C1 385	$[Hg_2(NS)_2]_x$ 1117	In <sub>2</sub> O <sub>3</sub> 863
HSO <sub>3</sub> F 177	HgO 299	In(OH) <sub>3</sub> 862
HSO <sub>3</sub> NH <sub>2</sub> 508	HgS 1111	InP 867
HSbCl <sub>6</sub> .4.5H <sub>2</sub> O 611	$Hg(SCN)_2$ 1123	InS 864
H <sub>2</sub> Se 418	$Hg_2(SCN)_2$ 1122	In <sub>2</sub> S 864
$H_2SeO_3$ 430	HgSe 1113	In <sub>2</sub> S <sub>3</sub> 864
$H_{2}SeO_{4}$ 432	HgSeF <sub>4</sub> 180	InSb 867
$H_2SiF_6$ 214		InSe 865
$H_2Si_2O_3$ 694, 699		In <sub>2</sub> Se 865
$H_2Si_2O_5$ 699	1	In <sub>2</sub> Se <sub>3</sub> 865
$H_4SiO_4$ 697	1 077	ln <sub>2</sub> Te 865
$H_2SnCl_6\cdot 6H_2O$ 730	$1_2 \ 277$	$\ln_2 \ln_3 865$
$H1aO_4 \cdot nH_2O = 1324$	$\begin{bmatrix} 10f & 291 \\ f(C, H, N) \end{bmatrix} C \begin{bmatrix} 0 & 297 \\ c \end{bmatrix} C \begin{bmatrix} 0 & 297 \\ c \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} C \begin{bmatrix} 0 & 0 \\ 0 & 0$	1r 1590
$H_2 Te 438$	$I(C_5H_5N)_2 C_1O_4 = 327$	IrCl <sub>3</sub> 1592
$H_2 IeU_3 449$	ICI 000	$1rF_4 = 271$
$H_{6} = 1006 + 451$	IC1 290	$1rF_{6} = 270$
$H(T_1C_1) > 2H_2O = 979$	$I(C_{1}O_{1})_{a} = 330$	$1rO_2$ 1590
$HT1(SO_{1}) = 4H_{1}O_{1} = 992$	$1(0104)_{3}$ 000	IrO 2·2H 20 1591
$HV(SO_4)_2 \cdot 4H_2O = 1282$	$IF_{7} = 160$	II 203-XII 20 1592
$H_{0} = W_{0} = 1423$	$I(IO_2)_2 = 331$	
$H_0 W \Omega_4 = 1423$	$I(NO_2)_2 = 329$	K
He $82$	$I_2O_4$ 333	
Hf 1172	$I_2O_5 = 307$	K 958
$HfBr_2$ 1204	I <sub>4</sub> O <sub>9</sub> 331	$K_2[Al_2O(OH)_6]$ 1693
$HfBr_3$ 1204	$I_2O_5 \cdot HIO_3 307$	K <sub>3</sub> AS 986
HfBr <sub>4</sub> 1203	I <sub>2</sub> O <sub>7</sub> .12MoO <sub>3</sub> .aq 1738	KAsH <sub>2</sub> 595
HfC 1245	$(IO)_2SO_4 \cdot H_2O$ 342	K[Au(CN) <sub>2</sub> ] 1065
HfCl <sub>4</sub> 1203	[IPyx]ClO4 328	KAuCl <sub>4</sub> .½H <sub>2</sub> O 1058
HfN 1233	[IPy <sub>2</sub> ]F 328	KAuO <sub>2</sub> .3H <sub>2</sub> O 1061
HfO <sub>2</sub> 1221	[IPy <sub>x</sub> ]NO <sub>3</sub> 328	KBF <sub>4</sub> 223
HfOCl <sub>2</sub> .8H <sub>2</sub> O 1213	I <sub>2</sub> (SO <sub>4</sub> ) <sub>3</sub> 329	KBF <sub>3</sub> OH 223

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 K<sub>3</sub>Bi 986 KBiO<sub>3</sub>.<sup>1</sup>/<sub>3</sub>H<sub>2</sub>O 628 KBrF<sub>4</sub> 237 KBrO.3H<sub>2</sub>O 311 KC<sub>8</sub> 635 KC<sub>24</sub> 635 KC<sub>36</sub> 635 KC48 635 K<sub>2</sub>CO<sub>3</sub> 987  $K_2Cd(CN)_4$  1106  $K_{3}[C_{0}(CN)_{6}]$  1541 K<sub>3</sub>CoF<sub>7</sub> 269  $K_3[Cr(CN)_6]$  1373  $\mathrm{K}_{3}[\mathrm{Cr}(\mathrm{C}_{2}\mathrm{O}_{4})_{3}]\cdot \mathrm{3H}_{2}\mathrm{O}$ 1372K<sub>2</sub>CrF<sub>6</sub> 269 K<sub>3</sub>CrO<sub>8</sub> 1391 K[CrO<sub>3</sub>Cl] 1390 K[CrO<sub>3</sub>F] 1390  $K_3[Cr(SCN)_6] \cdot 4H_2O$ 1374  $K[Cr(SCN)_4Py_2] \cdot 2H_2O$ 1379 K<sub>3</sub>CuF<sub>6</sub> 269 KCuO<sub>2</sub> 1014 KF 236 KF•HF 237 K<sub>2</sub>FeF<sub>6</sub> 269 K<sub>2</sub>FeO<sub>4</sub> 1504 KFeS<sub>2</sub> 1507 KGe 989 K<sub>2</sub>GeF<sub>6</sub> 216 KH 971 KHF<sub>2</sub> 146 KHMoO<sub>6</sub>·2H<sub>2</sub>O 1414 K7[HNb6O19.aq] 1706 KHPO<sub>3</sub>NH<sub>2</sub> 579 K<sub>2</sub>Hg(CN)<sub>4</sub> 1122 KHgI<sub>3</sub>·H<sub>2</sub>O 1110 K<sub>2</sub>Hg(SCN)<sub>4</sub> 1124 K<sub>2</sub>IrCl<sub>6</sub>-1593 K<sub>3</sub>IrCl<sub>6</sub>·3H<sub>2</sub>O 1595 KI 290 KIBr<sub>2</sub> 296 KICl<sub>2</sub> 295 KICl<sub>4</sub> 298

KIF<sub>6</sub> 238 KI<sub>3</sub>·H<sub>2</sub>O 294 KIO<sub>4</sub> 325 K<sub>3</sub>Mn(CN)<sub>6</sub> 1474  $K_5Mn(CN)_6$  1472  $K_4Mn(CN)_6.3H_2O$  1473  $K_{3}[Mn(C_{2}O_{4})_{3}].3H_{2}O$ 1470  $K_{2}[Mn(C_{2}O_{4})_{2}(OH)_{2}].$ ·2H<sub>2</sub>O 1471 K<sub>2</sub>MnCl<sub>5</sub> 1464 K<sub>2</sub>MnCl<sub>6</sub> 1464  $K_2MnF_6$  264,269 K<sub>2</sub>MnO<sub>4</sub> 1461  $K_4[M_0(CN)_8] \cdot 2H_2O$ 1416 K<sub>3</sub>MoCl<sub>6</sub> 1408 KN<sub>3</sub> 476 KNH<sub>2</sub> 1044  $K(NH_3)_2C_{12}$  636  $K_{3}[(NO)_{2}C_{0}(S_{2}O_{3})_{2}]$ 1766  $K[(NO)_2FeS_2O_3] \cdot H_2O$ 1766 $K_3[(NO)Mn(CN)_5]$ 1767  $K_4[(NO)M_0(CN)_5] \cdot H_2O$ 1766  $K_{3}[(NO)Ni(S_{2}O_{3})_{2}]$ 1766 K<sub>2</sub>NbF<sub>7</sub> 255 K<sub>3</sub>NbO<sub>8</sub> 1325 K<sub>3</sub>NbO<sub>8</sub>.<sup>1</sup>/<sub>2</sub>H<sub>2</sub>O 1325 K8[Nb6O19·aq] 1706 K<sub>2</sub>[Ni(CN)<sub>4</sub>]·H<sub>2</sub>O 1559 K<sub>2</sub>NiF<sub>6</sub> 269 KO<sub>2</sub> 981 K<sub>2</sub>O 974 3K2O·As2O5·24MoO3·aq 1734 K<sub>2</sub>O·3CrO<sub>3</sub> 1709 K<sub>2</sub>O·4CrO<sub>3</sub> 1710 3K20.P205.18M003.aq 1733  $11K_{2}O \cdot 2P_{2}O_{5} \cdot 24V_{2}O_{5}$ •aq 1740

 $3K_2O \cdot P_2O_5 \cdot 21WO_3 \cdot aq$ 17223K<sub>2</sub>O·P<sub>2</sub>O<sub>5</sub>·24WO<sub>3</sub>·aq 1721 $7K_2O \cdot P_2O_5 \cdot 22WO_3 \cdot aq$ 17222K<sub>2</sub>O·SiO<sub>2</sub>·12WO<sub>3</sub>·aq 1718 7K20.2SiO2.2OWO3.aq 1719  $K_2 O.3 V_2 O_5 1704$  $K_2OsO_4 \cdot 2H_2O$  1604  $K(OsO_3N)$  1605 KPF<sub>6</sub> 196 K<sub>3</sub>PO<sub>4</sub>.8H<sub>2</sub>O 545  $K_4P_2O_8$  562 K<sub>2</sub>PbCl<sub>6</sub> 753 KPbI<sub>3</sub> 754 KPbI3.2H2O 754 K<sub>2</sub>PdCl<sub>4</sub> 1584 K<sub>2</sub>PdCl<sub>6</sub> 1584 K<sub>2</sub>Pt(CN)<sub>4</sub>.3H<sub>2</sub>O 1576 K<sub>2</sub>PtCl<sub>4</sub> 1572 K<sub>2</sub>PtCl<sub>6</sub> 1571  $K_2Pt(OH)_6 \cdot xH_2O$  1575 K<sub>2</sub>ReCl<sub>6</sub> 1478 K<sub>2</sub>[RhCl<sub>5</sub>(H<sub>2</sub>O] 1588  $K_{3}RhCl_{6} H_{2}O$  1588 KRuO<sub>4</sub> 1600  $K_2RuO_4 \cdot H_2O$  1600 K<sub>2</sub>S 360 K<sub>2</sub>S<sub>2</sub> 363 K<sub>2</sub>S<sub>3</sub> 364 K<sub>2</sub>S<sub>4</sub> 366 K<sub>2</sub>S<sub>5</sub> 367 K<sub>2</sub>S<sub>6</sub> 368 KSCN 739 K<sub>2</sub>S<sub>2</sub>O<sub>8</sub> 392 K<sub>2</sub>S<sub>3</sub>O<sub>6</sub> 398 K<sub>2</sub>S<sub>3</sub>O<sub>10</sub> 1714 K<sub>2</sub>S<sub>4</sub>O<sub>6</sub> 399 K<sub>2</sub>S<sub>5</sub>O<sub>6</sub>·1½H<sub>2</sub>O 401  $K_2S_6O_6$  403 KSO<sub>2</sub>F 178  $K_2SO_3(NO)_2$  504 K<sub>3</sub>Sb 986, 1791

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 KSbCl<sub>6</sub>·H<sub>2</sub>O 612 K<sub>2</sub>Se 421 K<sub>2</sub>SeCl<sub>6</sub> 425 KSi 989 K<sub>2</sub>SnCl<sub>6</sub> 731 K<sub>2</sub>TaF<sub>7</sub> 256 K<sub>3</sub>TaO<sub>8</sub> 1325 K<sub>3</sub>TaO<sub>8</sub>.<sup>1</sup>/<sub>2</sub>H<sub>2</sub>O 1326 K<sub>2</sub>Te 441 K<sub>2</sub>TeCl<sub>6</sub> 444  $K_2(TlCl_5H_2O) \cdot H_2O$ 874 K<sub>3</sub>(TlCl<sub>6</sub>)·2H<sub>2</sub>O 873  $K_4[U(C_2O_4)_4] \cdot 5H_2O$ 1450 K<sub>2</sub>UO<sub>4</sub> 1445  $K_2VF_6$  269 K<sub>3</sub>V(SCN)<sub>6</sub> 1291  $KV(SO_4)_2$  1283  $K_{3}[W(CN)_{8}] \cdot H_{2}O = 1430$  $K_4[W(CN)_8] \cdot 2H_2O$ 1429  $K_3W_2Cl_9$  1427 K<sub>2</sub>Zn(CN)<sub>4</sub> 1088

## L

LaF<sub>3</sub> 246 La<sub>2</sub>S<sub>3</sub> 1153 La<sub>2</sub>Se<sub>3</sub> 1154 La<sub>2</sub>Se<sub>4</sub> 1154 Li 956 Li<sub>3</sub>Al 830 Li<sub>3</sub>AlAs<sub>2</sub> 831 LiAl(CN)<sub>4</sub> 833 LiAlH<sub>4</sub> 680, 805 Li<sub>3</sub>AlN<sub>2</sub> 828 Li<sub>3</sub>AlP<sub>2</sub> 830 Li<sub>3</sub>As 985 LiBH<sub>4</sub> 775  $LiBH_{4} \cdot O(C_{2}H_{5})_{2}$  775 Li<sub>3</sub>Bi 985 Li<sub>2</sub>C<sub>2</sub> 987 Li<sub>2</sub>CO<sub>3</sub> 950, 987

 $Li_3Cr(C_6H_5)_6$ .  $\cdot 2\frac{1}{2}(C_2H_5)_2O_{1375}$ LiF 235 LiFeO<sub>2</sub> 1504 LiGaH<sub>4</sub> 842 LiH 971,805 LiN<sub>3</sub> 475 Li<sub>3</sub>N 984 LiNH<sub>2</sub> 463 Li<sub>2</sub>NH 464 Li(NH<sub>3</sub>)<sub>2</sub>C<sub>12</sub> 636  $Li_2O$ Li<sub>2</sub>O<sub>2</sub> 975, 979 LiOH 983 LiOH H<sub>2</sub>O 983 Li<sub>3</sub>P 985 Li<sub>3</sub>Sb 985 Li<sub>2</sub>Si 991 Li<sub>4</sub>Si 991 Li<sub>2</sub>SiO<sub>3</sub> 705 LiUO<sub>3</sub> 1445 Li<sub>2</sub>UO<sub>4</sub> 1445 LnBr<sub>3</sub> 1148  $Ln(C_5H_5)_3$  1159 LnCl<sub>3</sub> 1146 LnI<sub>3</sub> 1149 LnN 1157 Ln(NO<sub>3</sub>)<sub>3</sub> 1158 Ln(OH)<sub>3</sub> 1152 LnS 1155  $Ln_2(SO_4)_3 \cdot nH_2O$  1156 LnSe 1155 LnTe 1155 LnX<sub>2</sub> 1150

### М

Mg 903 Mg3As<sub>2</sub> 917 MgBr<sub>2</sub> 909 MgC<sub>2</sub> 920 Mg<sub>2</sub>C<sub>3</sub> 920 MgCl<sub>2</sub> 905 MgCl<sub>2</sub>.6H<sub>2</sub>O 906 MgCl<sub>2</sub>·NH<sub>4</sub>Cl·6H<sub>2</sub>O 906  $M_{g}(ClO_{4})_{2}$  320  $M_{g}(ClO_{4})_{2} \cdot 6H_{2}O = 320$ MgF<sub>2</sub> 232 Mg<sub>2</sub>Ge 922  $MgH_2$  905 MgI<sub>2</sub> 910  $M_{g}(N_{3})_{2}$  917 Mg<sub>3</sub>N<sub>2</sub> 916 MgO 911 Mg(OD)<sub>2</sub> 137 Mg(OH)<sub>2</sub> 912  $[Mg(OH)_4]Na_2$  1683  $Mg_{3}P_{2}$  917 MgS 913  $Mg_3Sb_2$  606 MgSe 915 Mg<sub>2</sub>Si 921 MgTe 915 Mn 1454 Mn(CH<sub>3</sub>COO)<sub>3</sub> 1469  $Mn(CH_{3}COO)_{3} \cdot 2H_{2}O$ 1469  $[Mn(CN)_{5}(NO)]K_{3}$  1767 MnF<sub>2</sub> 262 MnF<sub>3</sub> 263 Mn<sub>4</sub>N 1468  $Mn_3[(NO)Mn(CN)_5]_2$ 1768 MnO 1455 MnO<sub>2</sub> 1458 Mn<sub>2</sub>O<sub>3</sub> 1457 Mn<sub>2</sub>O<sub>7</sub> 1459 Mn(OH)<sub>2</sub> 1456 MnO(OH) 1457 MnS 1465  $Mn_2(SO_4)_3$  1467 Mo 1401 MoBr<sub>3</sub> 1407  $[M_0Br_3Py_3]$  1408  $M_0(C_6H_6)_2$  1401  $[M_0(CN)_5(NO)]K_4 \cdot H_2O$ 1766 Mo(CO)<sub>6</sub> 1741 MoCl<sub>3</sub> 1404

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 MoCl<sub>5</sub> 1405 Mo<sub>3</sub>Cl<sub>6</sub> 1403 MoF<sub>6</sub> 259 MoO<sub>2</sub> 1409 MoO<sub>3</sub> 1412 Mo<sub>4</sub>O<sub>11</sub> 1410  $[M_0OCl_5](NH_4)_2$  1413 12MoO<sub>3</sub>.5(NH<sub>4</sub>)<sub>2</sub>O·aq 1711 $24M_0O_3 \cdot 3(NH_4)_2O \cdot P_2O_5 \cdot$ •aq 1730 Mo<sub>4</sub>O<sub>10</sub>(OH)<sub>2</sub> 1411 Mo5O5(OH)10 1411  $MoO_8[Zn(NH_3)_4]$  1414 MoS<sub>2</sub> 1415  $M_0S_4(NH_4)_2$  1416 MoSi<sub>2</sub> 1792, 1796

## N

N<sub>2</sub> 457 N<sub>3</sub>Br 477 NBr3.6NH3 480 NCl<sub>3</sub> 479 N<sub>3</sub>Cl 476 (NCl)<sub>3</sub>(SO)<sub>3</sub> 412  $NCl(SO_3K)_2$  508 ND<sub>3</sub> 137 NF3 181 NH<sub>3</sub> 460 <sup>15</sup>NH<sub>3</sub> 461 N<sub>2</sub>H<sub>4</sub> 469 N<sub>2</sub>H<sub>4</sub>·H<sub>2</sub>O 469  $NH_4AlF_4$  227 (NH<sub>4</sub>)<sub>3</sub>AlF<sub>6</sub> 226  $(NH_4)_3A_{sO_4}\cdot 3H_2O_6O_2$  $(NH_4)_3A_5S_4$  604 (NH<sub>4</sub>)<sub>2</sub>BeF<sub>4</sub> 232 NH<sub>2</sub>Br 480  $(NH_4)_2CS_3$  674 NH<sub>2</sub>Cl 477 NH<sub>4</sub>ClO<sub>3</sub> 313 (NH<sub>3</sub>)<sub>3</sub>CrO<sub>4</sub> 1392  $(NH_4)_2Cr_2O_{12} \cdot 2H_2O$ 1392

 $NH_4[Cr(SCN)_4(NH_3)_2]$ . ·H<sub>2</sub>O 1376 NH4F 183 NH<sub>4</sub>F•HF 183  $(NH_4)_4Fe(CN)_6$  1509  $(NH_4)_3GaF_6$  228  $NH_4Ga(SO_4)_2 \cdot 12H_2O$ 854 NH<sub>4</sub>HPO<sub>3</sub>NH<sub>2</sub> 584, 588 (NH<sub>4</sub>)<sub>2</sub>HPO<sub>3</sub>S 584 NH4HS 357  $(NH_4)_3InF_6$  229 (NH<sub>4</sub>)<sub>2</sub>IrCl<sub>6</sub> 1594 NH4I 289  $(NH_4)_2[M_0OC_{15}]$  1413 (NH<sub>4</sub>)<sub>2</sub>MoS<sub>4</sub> 1416 NH<sub>2</sub>NO<sub>2</sub> 496  $NH_4[(NO)_7Fe_4S_3] \cdot H_2O$ 1764  $3(NH_4)_2O \cdot As_2O_5 \cdot 24WO_3 \cdot$ •aq 1725 3(NH<sub>4</sub>)<sub>2</sub>O·Cr<sub>2</sub>O<sub>3</sub>·12MoO<sub>3</sub>· •aq 1737 NH<sub>2</sub>OH 501  $(NH_3OH)_3AsO_4$  501 (NH<sub>3</sub>OH)<sub>2</sub>C<sub>2</sub>O<sub>4</sub> 501 (NH<sub>3</sub>OH)Cl 498 (NH<sub>3</sub>OH)HSO<sub>4</sub> 499 (NH<sub>3</sub>OH)<sub>3</sub>PO<sub>4</sub> 500 5(NH4)20.12MoO3.aq 1711 $3(NH_4)_2O \cdot P_2O_5 \cdot 24M_0O_3 \cdot$ •aq 1730 5(NH<sub>4</sub>)<sub>2</sub>O·2P<sub>2</sub>O<sub>5</sub>· ·24V<sub>2</sub>O<sub>5</sub>·aq 1740 3(NH<sub>4</sub>)<sub>2</sub>O·P<sub>2</sub>O<sub>5</sub>·18WO<sub>3</sub>· •aq 1723  $3(NH_4)_2O \cdot P_2O_5 \cdot 24WO_3 \cdot$ •aq 1721 5(NH<sub>4</sub>)<sub>2</sub>O·12WO<sub>3</sub>·aq 1713 NH<sub>2</sub>OSO<sub>3</sub>H 511  $(NH_4)_2OsCl_6$  1603 NH<sub>4</sub>PF<sub>6</sub> 195  $(NH_4PO_3)_x$  580  $NH_4PO_2F_2$  196

 $(NH_4)_2P_2O_5(NH_2)_2$  588  $(NH_4)_2PbCl_6$  751 (NH<sub>4</sub>)<sub>2</sub>PdCl<sub>4</sub> 1584 (NH<sub>4</sub>)<sub>2</sub>PdCl<sub>6</sub> 1586  $(NH_4)_2 PtCl_6$  1570 NH<sub>4</sub>ReO<sub>4</sub> 1484  $(NH_4)_2[RhCl_5(H_2O)]$ 1588 (NH<sub>4</sub>)<sub>3</sub>RhCl<sub>6</sub>·H<sub>2</sub>O 1588  $(NH_4)_3[Rh(NO_2)_6]$  1586  $(NH_4)_2 RuCl_6$  1599  $(NH_4)_2S_5$  369  $NH_3SO_4$  510 (NH<sub>4</sub>)<sub>2</sub>S<sub>2</sub>O<sub>8</sub> 390 N<sub>2</sub>H<sub>6</sub>SO<sub>4</sub> 468 NH<sub>2</sub>SO<sub>3</sub>H 508  $NH(SO_3K)_2$  506 NH<sub>2</sub>SO<sub>3</sub>K 507  $N_2H_2(SO_3K)_2$  504, 509  $N_{2}H_{2}(SO_{3}NH_{4})_{2}$  509  $N_{2}H_{2}(SO_{3}Py)_{2}$  510 (NH<sub>4</sub>)<sub>2</sub>SbBr<sub>6</sub> 615 (NH<sub>4</sub>)<sub>2</sub>SeCl<sub>6</sub> 425  $(NH_4)_2SnCl_6$  731  $(NH_4)_2 TeCl_6$  444  $(NH_4)_2 TiCl_6$  1199  $(NH_4)_4[UO_2(CO_3)_3]$ 1449 NH<sub>4</sub>VO<sub>3</sub> 1272  $NH_4V(SO_4)_2$  1283  $NH_4V(SO_4)_2 \cdot 12H_2O$ 1284 $(NH_4)_2 ZnCl_4 1072$  $(NH_4)_2 Zn(SO_4)_2 \cdot 6H_2O$ 1077 NHg<sub>2</sub>Br 1117  $NHg_2OH \cdot xH_2O$  1116 NI<sub>3</sub>.NH<sub>3</sub> 481 NO 485 NO<sub>2</sub> 488 N<sub>2</sub>O 484 N<sub>2</sub>O<sub>3</sub> 487 N<sub>2</sub>O<sub>4</sub> 488 N<sub>2</sub>O<sub>5</sub> 489  $NOBF_4$  224 NOBF4·H2O 224

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 NO<sub>2</sub>[BF<sub>4</sub>] 187 N<sub>2</sub>O<sub>5</sub>.BF<sub>3</sub> 187 NOBr 513 NOC1 511 NO<sub>2</sub>Cl 513 NOClO<sub>4</sub> 320  $NO_2CIO_4$  321 NOF 184 NO<sub>2</sub>F 186 NO<sub>3</sub>F 187  $[(NO)_2 FeSC_2 H_5]_2$  1765 (NO)HSO<sub>4</sub> 406  $NOH(SO_3K)_2$  503 NO<sub>2</sub>NH<sub>2</sub> 497 NO<sub>2</sub>NHCOOK 497 NOSO<sub>2</sub>F 186 NO(SO<sub>3</sub>K)<sub>2</sub> 504  $NO(SbCl_6)$  612  $NO_2(SbCl_6)$  612  $N_2(SO_3K)_2$  510  $N(SO_{3}K)_{3} \cdot 2H_{2}O = 506$ Na 958 Na<sub>3</sub>AgO<sub>2</sub> 1039 NaAlCl<sub>4</sub> 816  $Na_4[Al(OH)_7] \cdot 3H_2O$  1692  $Na_4[Al_4O_3(OH)_{10}]$  1693  $Na_{6}[Al_{6}O_{4}(OH)_{16}]$  1693 Na<sub>3</sub>As 986, 1791, 1793 NaAsH<sub>2</sub> 595  $Na_3AsO_2S_2 \cdot 11H_2O$  605  $Na_3AsO_3S \cdot 12H_2O = 605$  $Na_3AsS_4 \cdot 8H_2O$  604 Na<sub>2</sub>Au 1793  $Na[B(C_6H_5)_4]$  803 NaBF<sub>4</sub> 222 NaBH<sub>4</sub> 776 NaBH<sub>4</sub>·2H<sub>2</sub>O 777 NaBO<sub>2</sub> 791, 793 NaBO<sub>2</sub>•½H<sub>2</sub>O 791 NaBO2·2H2O 791  $NaBO_2 \cdot 4H_2O$  791  $NaBO_2 \cdot H_2O_2 \cdot 3H_2O$  796 NaBO<sub>3</sub>.4H<sub>2</sub>O 795  $NaB_{5}O_{8} \cdot 5H_{2}O_{795}$ Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub> 794 Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub>·2H<sub>2</sub>O 794

 $Na_{2}B_{4}O_{7} \cdot 4H_{2}O_{7}94$ Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub>.5H<sub>2</sub>O 793 Na<sub>2</sub>B<sub>4</sub>O<sub>7</sub>·10H<sub>2</sub>O 793 Na<sub>3</sub>BO<sub>3</sub> 790 Na4BeO3 895 Na<sub>3</sub>Bi 986, 1793 NaBiO<sub>3</sub> 627 NaBiO3-nH2O 628 NaBrO.5H<sub>2</sub>O 310 Na<sub>2</sub>C<sub>2</sub> 987  $Na_2CO_3$  988  $Na_{3}[(CO)Fe(CN)_{5}]$  1769 NaClO 309 NaClO.5H<sub>2</sub>O 309 NaClO2.3H2O 312 Na<sub>3</sub>[Co(NO<sub>2</sub>)<sub>6</sub>] 1541  $Na_3[Cr(OH)_6]$  1688 NaCrS<sub>2</sub> 1394  $Na_{2}[Cu(OH)_{4}]$  1684 NaF 235  $Na_2[Fe(CN)_5NH_3] \cdot H_2O$ 1512Na<sub>3</sub>[Fe(CN)<sub>5</sub>NH<sub>3</sub>].3H<sub>2</sub>O 1511 Na<sub>2</sub>[Fe(OH)<sub>4</sub>] 1686 Na<sub>3</sub>[Fe(OH<sub>2</sub>)(CN)<sub>5</sub>] 1770  $Na_4[Fe(OH)_7] \cdot 2H_2O$ 1689  $Na_5[Fe(OH)_8] \cdot 5H_2O$ 1690  $Na_3Fe(SCN)_6 \cdot 12H_2O$ 1511NaGe 989 NaGeH<sub>3</sub> 714 Na<sub>2</sub>GeH<sub>2</sub> 714 NaH 971  $NaH_2AsO_4 \cdot H_2O$  602 Na<sub>3</sub>H<sub>2</sub>As<sub>3</sub>O<sub>10</sub> 1709 NaHB(OCH<sub>3</sub>)<sub>3</sub> 777 Na<sub>3</sub>H<sub>2</sub>IO<sub>6</sub> 323 Na7[HNb6O19.aq] 1706 Na 3HP 207 548 NaH<sub>2</sub>PO<sub>4</sub>.2H<sub>2</sub>O 544  $Na_{2}H_{2}P_{2}O_{6}.6H_{2}O$  560 NaHS 357 NaHSe 419

Na<sub>2</sub>H<sub>4</sub>TeO<sub>6</sub> 453  $Na_2H_4TeO_6 \cdot 3H_2O$  454 NaIO<sub>3</sub> 323 NaIO<sub>4</sub> 323, 324 NaIO<sub>4</sub>.3H<sub>2</sub>O 324  $Na_{2}[Mg(OH)_{4}]$  1683 Na<sub>3</sub>MnO<sub>4</sub>. 4NaOH · 12H<sub>2</sub>O 1460 NaN<sub>3</sub> 474 Na<sup>15</sup>N<sub>3</sub> 466 NaNH<sub>2</sub> 465 Na<sup>15</sup>NH<sub>2</sub> 466  $Na(NH_3)_2C_{12}$  636 NaNO 514 Na<sub>2</sub>NO<sub>2</sub> 515  $Na_2N_2O_3$  517  $Na_2N_2O_2.9H_2O_495$  $Na_2[(NO)Fe(CN)_5] \cdot 2H_2O$ 1768  $Na[(NO)_2FeS].4H_2O$ 1763  $Na[(NO)_2FeS_2O_3]$  1766 Na<sub>8</sub>[Nb<sub>6</sub>O<sub>19</sub>·aq] 1706 NaO<sub>2</sub> 980 Na<sub>2</sub>O 974 Na<sub>2</sub>O<sub>2</sub> 979 Na20.As205.6MoO3.aq 1736 3Na<sub>2</sub>O·As<sub>2</sub>O<sub>5</sub>·18MoO<sub>3</sub>· •aq 1735  $5Na_2O \cdot B_2O_3 \cdot 24WO_3 \cdot aq$ 1717 NaOD 121 5Na20.I207.12MoO3.aq 1738 Na<sub>2</sub>O•4MoO<sub>3</sub>•aq 1712 5Na<sub>2</sub>O·12MoO<sub>3</sub>·aq 1710  $10Na_2O \cdot P_2O_5 \cdot 24V_2O_5 \cdot aq$ 1739 3Na<sub>2</sub>O·P<sub>2</sub>O<sub>5</sub>·24WO<sub>3</sub>·aq 17202Na2O·SiO2·12MoO3·aq 1729 Na<sub>2</sub>O·V<sub>2</sub>O<sub>5</sub>·aq 1703 2Na<sub>2</sub>O·V<sub>2</sub>O<sub>5</sub>·aq 1702 3Na<sub>2</sub>O·5V<sub>2</sub>O<sub>5</sub>·aq 1703

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Na<sub>2</sub>O·4WO<sub>3</sub>·aq 1727 5Na<sub>2</sub>O·12WO<sub>3</sub>·aq 1712  $Na_2OsCl_6 \cdot 2H_2O$  1602 Na<sub>3</sub>P 986 NaPH<sub>2</sub> 530 NaPH2.2NH3 530 Na<sub>2</sub>PH 530  $(NaPO_3)_x$  549  $(NaPO_3)_v$  550 (NaPO<sub>3</sub>)<sub>z</sub> 551 Na<sub>3</sub>P<sub>3</sub>O<sub>9</sub>·6H<sub>2</sub>O 552  $Na_4P_2O_6 \cdot 10H_2O_561$  $Na_4P_4O_{12} \cdot nH_2O$  553  $Na_5P_3O_{10}$  547 Na<sub>5</sub>P<sub>3</sub>O<sub>10</sub>.6H<sub>2</sub>O 547 Na<sub>6</sub>P<sub>4</sub>O<sub>13</sub> 548  $(Na_2PO_3)_3N 590$  $Na_2PO_3NH_2$  588  $Na_2PO_3NH_2.6H_2O$  581 Na<sub>4</sub>P<sub>2</sub>O<sub>6</sub>NH 589  $Na_4P_2O_6NH \cdot 10H_2O_589$ Na<sub>3</sub>POS<sub>3</sub>·11H<sub>2</sub>O 571  $Na_3PO_2S_2 \cdot 11H_2O 570$ Na<sub>3</sub>PO<sub>3</sub>S·12H<sub>2</sub>O 569  $NaPOS(NH_2)_2$  589 Na<sub>3</sub>PS<sub>4</sub>.8H<sub>2</sub>O 572 Na<sub>15</sub>Pb<sub>4</sub> 1793 Na<sub>2</sub>PbO<sub>3</sub> 758 Na<sub>4</sub>PbO<sub>4</sub> 759  $Na_{2}[Pb(OH)_{6}]$  1694 Na<sub>2</sub>PdCl<sub>4</sub> 1584 Na<sub>2</sub>PtCl<sub>6</sub> 1571 Na<sub>2</sub>PtCl<sub>6</sub>.6H<sub>2</sub>O 1571  $Na_2Pt(OH)_6 \cdot xH_2O = 1575$ Na2ReO3 1483 Na<sub>3</sub>RhCl<sub>6</sub>·12H<sub>2</sub>O 1587 Na<sub>2</sub>S 358 Na<sub>2</sub>S<sub>2</sub> 361 Na<sub>2</sub>S<sub>4</sub> 365 Na<sub>2</sub>S<sub>5</sub> 367  $Na_2S_2O_4 \cdot 2H_2O_393$  $Na_2S_2O_6 \cdot 2H_2O_395$ Na<sub>3</sub>Sb 986, 1793 Na<sub>3</sub>SbS<sub>4</sub>.9H<sub>2</sub>O 619 Na<sub>2</sub>Se 421 Na<sub>2</sub>Se<sub>2</sub> 421

Na<sub>2</sub>SeO<sub>3</sub>.5H<sub>2</sub>O 431 Na<sub>2</sub>SeO<sub>4</sub> 433  $Na_2SeS_4O_6 \cdot 3H_2O$  434 NaSi 989 Na<sub>2</sub>SiO<sub>3</sub> 704 Na<sub>2</sub>SiO<sub>3</sub>.9H<sub>2</sub>O 704 Na2Si2O5 704 Na<sub>15</sub>Sn<sub>4</sub> 1793  $Na_4SnO_4$  739  $Na[Sn(OH)_{3}]$  1687 Na<sub>2</sub>[Sn(OH)<sub>6</sub>] 1694 Na<sub>2</sub>SnS<sub>3</sub>.8H<sub>2</sub>O 742  $Na_4SnS_4 \cdot 18H_2O$  743 Na7[Ta5O16·aq] 1708 Na8[Ta6O19.aq] 1708 Na<sub>2</sub>Te 441 Na<sub>2</sub>Te<sub>2</sub> 442 Na<sub>2</sub>TeO<sub>3</sub> 449 Na<sub>2</sub>TeO<sub>4</sub> 449, 453 Na6TeO6 453 Na2TeS4O6·2H2O 454 NaUO<sub>3</sub> 1445 NaZn<sub>13</sub> 1793 Na[Zn(OH)3] 1681 Na<sub>2</sub>[Zn(OH)<sub>4</sub>] 1682 NbBr<sub>2</sub> 1309 NbBr<sub>3</sub> 1309 NbBr5 1311 NbC 1333 NbCl<sub>2</sub> 1296 NbCl<sub>3</sub> 1297 NbCl<sub>4</sub> 1299 NbCl<sub>5</sub> 1302 NbF<sub>5</sub> 254 NbH 1296 NЫ<sub>2</sub> 1314 NbI<sub>3</sub> 1314 NbI<sub>4</sub> 1314 NbI<sub>5</sub> 1315 Nb--N 1328 NbO 1317 NbO<sub>2</sub> 1318 Nb<sub>2</sub>O<sub>5</sub> 1318 NbOBr<sub>3</sub> 1313 NbOCl<sub>3</sub> 1307 [Nb<sub>6</sub>O<sub>19</sub>.aq]Nag 1705

NbP 1330 NbP<sub>2</sub> 1330 Nb-S 1327 NbSi<sub>2</sub> 1792, 1797 Ne 82 Ni 1543 NiBr<sub>2</sub> 1545 NiBr<sub>2</sub>.6H<sub>2</sub>O 1546 Ni<sub>3</sub>C 1556 NiCO<sub>3</sub> 1557 NiCO<sub>3</sub>.6H<sub>2</sub>O 1556 Ni(CO)<sub>4</sub> 1747 Ni(CO)<sub>2</sub>C<sub>12</sub>H<sub>8</sub>N<sub>2</sub> 1750 NiCl<sub>2</sub> 1544 NiF<sub>2</sub> 269 NiI<sub>2</sub> 1547 NiI<sub>2</sub>.6H<sub>2</sub>O 1547 Ni<sub>3</sub>N 1555 Ni<sub>3</sub>N<sub>2</sub> 1555  $Ni(NH_2)_2$  1554 [Ni(NH<sub>3</sub>)<sub>6</sub>]Cl<sub>2</sub> 1545  $[Ni(NO)(S_2O_3)_2]K_3$  1766 NiO 1548 Ni(OH)<sub>2</sub> 1549  $[Ni(OH)_6]Sr_2$  1686 NiO(OH) 1549 Ni<sub>3</sub>O<sub>2</sub>(OH)<sub>4</sub> 1551 NiS 1551 NiS<sub>2</sub> 1554 Ni(SCN)<sub>2</sub> 1558  $NiS_2(C_6H_5 \cdot CSS)_2Ni(SSC \cdot$  $\cdot C_6H_5)_2$  1558

## 0

 $\begin{array}{c} O_2 \ \ 334 \\ O_3 \ \ 337 \\ OF_2 \ \ 163 \\ O_2F_2 \ \ 162 \\ [OS(N)Cl]_3 \ \ 412 \\ Os \ \ 1601 \\ OsCl_4 \ \ 1601 \\ OsO_2 \ \ 1603 \\ OsO_4 \ \ 1604 \end{array}$ 

### 1822

pp 1-992: Vol. 1; pp 1003-18	810: Vol. 2	
Р	PS(NH <sub>2</sub> ) <sub>3</sub> 587	RbC 24 635
	PaSe3 573	RbC 36 635
P 518	Pb 748	RbC48 635
PC1 <sub>2</sub> F 191	Pb(CH3)4 763	Rb <sub>2</sub> CO <sub>3</sub> 987
PC1 <sub>2</sub> F <sub>3</sub> 192	$Pb(C_2H_5)$ 765	RbC1 951
$PCl_{4}$ $PF_{6}$ 193	$Pb(CH_2COO)_4$ 767	Rb2CrO₄ 1388
PF <sub>3</sub> 179, 189	$PbCO_2$ 766	Rb2Cr2O7 1388
PF <sub>5</sub> 190, 194	$2PbCO_3 \cdot Pb(OH)_2 = 767$	RbGe 989
PH <sub>3</sub> 525	$PbCl_{4}$ 750	RbH 971
$P_{2}H_{4}$ 525	$PbF_2$ 218	RbN 3 476
PH₄I 531	$PbF_{4}$ 219	Rb(NH <sub>3</sub> ) <sub>2</sub> C <sub>12</sub> 637
PI <sub>3</sub> 540	$Pb(N_3)_2$ 763	RbO <sub>2</sub> 981
P <sub>2</sub> I <sub>4</sub> 539	PbO <sub>2</sub> 757, 1668	Rb <sub>2</sub> O 974
$P_{3N5}$ 574	Pb304 755	<b>RЬ́ОН 983</b>
$(PNBr_2)_{n}$ 578	$[Pb(OH)_6]Na_2$ 1694	2Rb2O·SiO2·12MoO3·ag
$(PNCl_2)_{n} 575$	$Pb_2P_2Q_6$ 558	1730
$(PNF_2)_3 = 194$	PbS 760	Rb2S2 369
$(PNF_2)_{A}$ 194	Pb(SCN) <sub>2</sub> 769	Rb <sub>2</sub> S <sub>3</sub> 369
$P_{2}O_{5}$ 541.825	$Pb(SO_4)_2$ 761	Rb <sub>2</sub> S <sub>5</sub> 369
POBr <sub>3</sub> 534	$PbSiO_3$ 705	$Rb_2SeCl_6$ 425
P2O3Cl4 536	Pd 1580	RbŠi 989
$P_4O_4Cl_{10}$ 536	[PdBr2(NH3)2] 1585	Rb <sub>2</sub> TeCl <sub>6</sub> 444
$POCI(OC_6H_5)_2$ 579	PdCl <sub>2</sub> 1582	$Rb(TlBr_4) \cdot H_2O = 876$
$POCl_2(OC_6H_5)$ 582	$[PdCl_{2}(NH_{3})_{2}]$ 1585	Rb <sub>3</sub> (T1Br <sub>6</sub> ). <sup>3</sup> H <sub>2</sub> O 876
POF <sub>3</sub> 179, 193	[PdCl4]Na2 1584	Re 1476
$PO(NH_2)_3 584$	PdO 1583	ReCl <sub>3</sub> 1476
$P_{2}O_{3}(NH_{2})_{4}$ 588	PrO <sub>2</sub> 1151	ReCl <sub>5</sub> 1477
$[PO(NH)NH_2]_{n}$ 588	Pt 1560	ReF <sub>6</sub> 264
$[PO(NH_2)_2]NH 587$	PtCl <sub>2</sub> 1568	ReO <sub>2</sub> 1480
[PO(NH <sub>2</sub> ) <sub>2</sub> NH] <sub>2</sub> PONH <sub>2</sub>	PtCl <sub>3</sub> 1567	ReO <sub>3</sub> 1481
587	PtCl <sub>4</sub> 1567	Re <sub>2</sub> O <sub>7</sub> 1482
$PONH_2(OC_6H_5)_2$ 577	[PtCl <sub>2</sub> (NH <sub>3</sub> ) <sub>2</sub> ] 1578	ReOC14 1479
$PO(NH_2)_2(OC_6H_5)$ 582	[PtCl <sub>6</sub> ]Na <sub>2</sub> 1571	ReO <sub>3</sub> Cl 1480
P2O5.18MoO3.aq 1732	[Pt(NH <sub>3</sub> ) <sub>4</sub> ]Cl <sub>2</sub> ·H <sub>2</sub> O 1577	ReS <sub>2</sub> 1486
P2O5.24MoO3.aq 1731	$[Pt(NH_3)_4][PtCl_4]$ 1577	Re <sub>2</sub> S <sub>7</sub> 1487
P205.18WO3.aq 1724	$[Pt(NO_2)_2(NH_3)_2]$ 1579	Rh 1585
P2O5.24WO3.aq 1720	PtO 1573	RhCl <sub>3</sub> 1587
$P_2S_5$ 567	$PtO_2 \cdot xH_2O$ 1574	[RhC1(NH <sub>3</sub> ) <sub>5</sub> ]C1 <sub>2</sub> 1590
$P_2S_5 \cdot 7NH_3  574 \sim$	[Pt(OH) <sub>6</sub> ]Na <sub>2</sub> ·xH <sub>2</sub> O 1575	[RhC1(NH <sub>3</sub> ) <sub>5</sub> ](NO <sub>3</sub> ) <sub>2</sub>
P <sub>4</sub> S <sub>3</sub> 563	PtS 1575	1590
$P_{4}S_{5}$ 565	PtS <sub>2</sub> 1576	$[RhCl_{5}(H_{2}O)](NH_{4})_{2}$
P <sub>4</sub> S <sub>7</sub> 566		1588
P <sub>4</sub> S <sub>10</sub> 567	R	[RhCl <sub>6</sub> ](NH <sub>4</sub> ) <sub>3</sub> ·H <sub>2</sub> O
PSBr <sub>3</sub> 535		1586
PSBr <sub>3</sub> ·H <sub>2</sub> O 536	Rь 958	[RhCl <sub>6</sub> ]Na <sub>3</sub> .12H <sub>2</sub> O 1587
PSC1 <sub>3</sub> 532	RbC <sub>8</sub> 635	$[Rh(NO_2)_6](NH_4)_3$ 1586

S

S 341 S<sub>2</sub>Br<sub>2</sub> 377 S<sub>3</sub>Br<sub>2</sub> 379 S<sub>4</sub>Br<sub>2</sub> 379 S<sub>5</sub>Br<sub>2</sub> 379 S<sub>6</sub>Br<sub>2</sub> 379 S<sub>7</sub>Br<sub>2</sub> 379 S<sub>8</sub>Br<sub>2</sub> 376 (SCN)<sub>2</sub> 671 SCl<sub>2</sub> 370 SCl<sub>4</sub> 376  $S_2Cl_2$  371 S<sub>3</sub>Cl<sub>2</sub> 373 S<sub>4</sub>Cl<sub>2</sub> 372, 375 S<sub>5</sub>Cl<sub>2</sub> 372, 375 S<sub>6</sub>Cl<sub>2</sub> 372, 375 S<sub>7</sub>Cl<sub>2</sub> 372, 376 S<sub>8</sub>Cl<sub>2</sub> 372, 376 SF4 168 SF<sub>6</sub> 169 S<sub>2</sub>N<sub>2</sub> 409 S<sub>4</sub>N<sub>2</sub> 408 S<sub>4</sub>N<sub>4</sub> 406  $S_4(NH)_4$  411 S<sub>7</sub>NH 411  $S_{3}N_{2}O_{2}$  413  $S_{3}N_{2}O_{5}$  414 SO 379  $(SO_{3-4})_x$  382 S<sub>2</sub>O 380 S<sub>2</sub>O<sub>3</sub> 380 SOBr<sub>2</sub> 387 SO<sub>2</sub>BrF 176

SOC1<sub>2</sub> 382 SO<sub>2</sub>Cl<sub>2</sub> 383 S<sub>2</sub>O<sub>5</sub>Cl<sub>2</sub> 386 SOCIF 174 SO<sub>2</sub>ClF 175 SOF<sub>2</sub> 170, 179 SOF<sub>4</sub> 171  $SO_2F_2$  173  $S_{3}O_{8}F_{2}$  174  $(SO_2N)_3Ag_3 \cdot 3H_2O 483$ SONH 480  $(SO_2NH)_3$  483  $SO_2(NH_2)_2$  482  $SO_2(NHAg)_2$  483 Sb 606 SbBr<sub>3</sub> 613 SbCl<sub>3</sub> 608 SbCl<sub>5</sub> 610 SbCl<sub>5</sub>.H<sub>2</sub>O 610 SbCl<sub>5</sub>·4H<sub>2</sub>O 610 SbCl<sub>2</sub>F<sub>3</sub> 200 SbF<sub>3</sub> 199 SbF<sub>5</sub> 143, 200 SbH<sub>3</sub> 606 SbI3 614 Sb<sub>2</sub>O<sub>3</sub> 615 Sb<sub>2</sub>O<sub>4</sub> 618 Sb<sub>2</sub>O<sub>5</sub> 616  $Sb_2O_5 (H_2O)_x 617$ SbOC1 611 Sb<sub>4</sub>O<sub>5</sub>Cl<sub>2</sub> 611 (SbO)<sub>2</sub>SO<sub>4</sub> 619  $Sb_2(SO_4)_3$  618 ScF<sub>3</sub> 245 Se 415 SeBr<sub>4</sub> 427 Se<sub>2</sub>Br<sub>2</sub> 426 SeCl<sub>4</sub> 423 Se<sub>2</sub>Cl<sub>2</sub> 422 SeF<sub>4</sub> 180 SeF<sub>6</sub> 179 Se<sub>4</sub>N<sub>4</sub> 435 SeO<sub>2</sub> 428 SeOCl<sub>2</sub> 429  $SeO(OC_2H_5)_2$  435 SeSO<sub>3</sub> 435

Si 676 SiBr<sub>2</sub> 687 SiBr<sub>4</sub> 686, 688 Si<sub>2</sub>Br<sub>6</sub> 688 Si(CH<sub>3</sub>COO)<sub>4</sub> 701 SiCH<sub>3</sub>Cl<sub>3</sub> 695 Si(CH<sub>3</sub>)<sub>2</sub>Cl<sub>2</sub> 694, 695 SiCl<sub>4</sub> 680, 682 Si<sub>2</sub>Cl<sub>6</sub> 680 Si<sub>3</sub>Cl<sub>8</sub> 684 Si<sub>4</sub>Cl<sub>10</sub> 684 Si5Cl12 684 Si<sub>6</sub>Cl<sub>14</sub> 684 Si10Cl22 684 Si<sub>10</sub>Cl<sub>20</sub>H<sub>2</sub> 685 SiF<sub>4</sub> 212 (SiH)<sub>x</sub> 681  $(SiH_2)_{x}$  681 SiH<sub>4</sub> 679, 680 Si<sub>2</sub>H<sub>6</sub> 679 Si<sub>3</sub>H<sub>8</sub> 679 SiHBr<sub>3</sub> 692  $SiH_2Br_2$  694 SiHCl<sub>3</sub> 691 SiH<sub>2</sub>Cl<sub>2</sub> 691 SiH<sub>3</sub>Cl 691 SiHF<sub>3</sub> 214 SiI<sub>4</sub> 689 Si<sub>2</sub>I<sub>6</sub> 690 Si(N<sub>3</sub>)<sub>4</sub> 476 Si(NCO)<sub>4</sub> 702 SiO 696 Si<sub>2</sub>O<sub>3</sub> 700 Si(OCH<sub>3</sub>)<sub>4</sub> 702  $Si(OC_2H_5)_4$  702 Si(OCN)<sub>4</sub> 702 Si<sub>2</sub>OCl<sub>6</sub> 696 Si<sub>3</sub>O<sub>2</sub>Cl<sub>8</sub> 696 Si<sub>4</sub>O<sub>3</sub>Cl<sub>10</sub> 696 Si<sub>4</sub>O<sub>4</sub>Cl<sub>8</sub> 695 Si504Cl12 696 Si<sub>6</sub>O<sub>5</sub>Cl<sub>14</sub> 696 Si<sub>7</sub>O<sub>6</sub>Cl<sub>16</sub> 696 SiO<sub>2</sub>.12WO<sub>3</sub>.aq 1718 SiS 700 SiS<sub>2</sub> 700

$\begin{array}{cccccccccccccccccccccccccccccccccccc$
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$\begin{array}{cccccccccccccccccccccccccccccccccccc$
Sn 727         TaBr <sub>4</sub> 1310         TiCl <sub>3</sub> 1187           SnBr <sub>2</sub> 732         TaBr <sub>5</sub> 1311         TiCl <sub>3</sub> .6H <sub>2</sub> O 1187, 1193
SnBr <sub>2</sub> 732 TaBr <sub>5</sub> 1311 TiCl <sub>3</sub> .6H <sub>2</sub> O 1187, 1193
SnBr <sub>4</sub> 733 TaC 1331 TiCl <sub>4</sub> 1195
$Sn(CH_3)_4$ 744 TaCl <sub>4</sub> 1301 TiF <sub>3</sub> 248
$Sn(C_{2}H_{5})_{4}$ 746 TaCl <sub>5</sub> 1302, 1305 TiF <sub>4</sub> 250
$Sn(CH_3COO)_4$ 747 TaF <sub>5</sub> 255 Ti-H 1184
SnCl <sub>2</sub> 728 Ta-H 1296 TiH <sub>2</sub> 114
SnCl <sub>4</sub> 729 Tal <sub>5</sub> 1316 Til <sub>2</sub> 1185
SnF <sub>2</sub> 217 Ta-N 1328 Til <sub>3</sub> 1187, 1192
$SnF_4$ 217 $Ta_2O_5$ 1318 $TiI_4$ 1205
SnI <sub>2</sub> 734 [Ta <sub>5</sub> O <sub>16</sub> .aq]Na <sub>7</sub> 1708 TiN 1233
$SnI_4$ 735 [Ta <sub>6</sub> O <sub>19</sub> ag]Na <sub>8</sub> 1708 Ti(NO <sub>3</sub> ) <sub>4</sub> 1237
SnO 736 TaP 1330 TiO 1214
$S_nO_2$ 738 $T_aP_2$ 1330 $T_iO_2$ 1216
$S_n O_{2} \cdot nH_2 O_{737}$ Ta-S 1328 TiO <sub>2</sub> $\cdot nH_2 O_{1218}$
$[Sn(OH)_2]N_8$ 1687 TaSi <sub>2</sub> 1792, 1797 Ti <sub>2</sub> O <sub>3</sub> 1214
[Sn(OH) <sub>6</sub> ]Na <sub>2</sub> 1694 Te 437 TiOCl 1209
$S_nS_739$ TaBr <sub>4</sub> 445 TiOCl <sub>2</sub> 1209
$SnS_2$ 741 $TeCl_4$ 442 $TiO(NO_3)_2$ 1241
$Sn(SO_4)_2 \cdot 2H_2O_744$ TeF <sub>6</sub> 180 TiOSO <sub>4</sub> 1228
$S_{r}$ 922 $TeI_{4}$ 447 $TiOSO_{4} \cdot 2H_{2}O$ 1229
SrBr <sub>2</sub> 930 TeO <sub>2</sub> 447 TiP 1241
$SrCO_3$ 931 $TeO_3$ 450 $TiS_2$ 1222
SrCl <sub>2</sub> 930 Te <sub>2</sub> O <sub>3</sub> (OH)NO <sub>3</sub> 437, 447 TiS <sub>&lt;2</sub> 1222, 1224
$Sr(C10_4)_2$ 320 TeSO <sub>3</sub> 455 TiS <sub>3</sub> 1222
$Sr(C10_4)_2 \cdot 4H_2O$ Th 1174 $Ti_2(SO_4)_3 \cdot 1226$
320 ThBr <sub>4</sub> 1203 TiSi <sub>2</sub> 1249
SrF <sub>2</sub> 234 ThC 1248 TI 867
SrH <sub>2</sub> 929 ThC <sub>2</sub> 1248 T1Br 869
Srl <sub>2</sub> 930 ThCl <sub>4</sub> 1203 TlBr <sub>3</sub> ·4H <sub>2</sub> O 874
$Sr(N_3)_2$ 941 ThCl <sub>4</sub> ·8H <sub>2</sub> O 1204 Tl <sub>2</sub> CO <sub>3</sub> 884
Sr <sub>3</sub> N <sub>2</sub> 940 Th-H 1185 TlC1 869
Sr <sub>2</sub> [Ni(OH) <sub>6</sub> ] 1686 ThI <sub>4</sub> 1205 TlCl <sub>3</sub> 870
SrO 932 Th <sub>3</sub> N <sub>4</sub> 1236 TlCl <sub>3</sub> ·4H <sub>2</sub> O 870
SrO <sub>2</sub> 936 Th(NO <sub>3</sub> ) <sub>4</sub> 1240 T1F 230
SrO <sub>2</sub> ·8H <sub>2</sub> O 935 Th(NO <sub>3</sub> ) <sub>4</sub> ·nH <sub>2</sub> O 1238 TlF <sub>3</sub> 230
Sr(OH) <sub>2</sub> ·8H <sub>2</sub> O 935 ThO <sub>2</sub> 1221 TII 869
SrS 938 Th <sub>3</sub> P <sub>4</sub> 1241 Tll·I <sub>2</sub> 876
SrSe 939 ThSi <sub>2</sub> 1249, 1792, 1797 TH <sub>3</sub> 876
SrSeO <sub>3</sub> 939 Ti 1161 Tl <sub>3</sub> N 883
SrSeO <sub>4</sub> 939 TiBr <sub>2</sub> 1185 TINO <sub>3</sub> 883
SrSi 947 TiBr <sub>3</sub> 1187, 1192 Tl <sub>2</sub> O 877
SrTe 940 TiBr <sub>3</sub> ·6H <sub>2</sub> O 1187, 1195 Tl <sub>2</sub> O <sub>3</sub> 879
# U

U 1431 UBr<sub>4</sub> 1440 U(C<sub>2</sub>O<sub>4</sub>)<sub>2</sub>.6H<sub>2</sub>O 1449 UCl<sub>3</sub> 1435 UCl<sub>4</sub> 1436 UCl<sub>5</sub> 1438  $UD_3$  123 UF<sub>4</sub> 261 UF<sub>6</sub> 262 UH<sub>3</sub> 113, 1434 UO<sub>2</sub> 1442 UO<sub>3</sub> 1442 U(OC<sub>2</sub>H<sub>5</sub>)<sub>5</sub> 1451  $U(OC_2H_5)_6$  1452  $UO_2(C_{15}H_{11}O_2)_2$  1453 UO<sub>2</sub>Cl<sub>2</sub> 1439 UO<sub>4</sub>·2H<sub>2</sub>O 1446 US<sub>2</sub> 1446  $U(SO_4)_2 \cdot 4H_2O = 1447$  $U(SO_4)_2 \cdot 8H_2O = 1447$ USi<sub>2</sub> 1792

# V

V 1252 VBr<sub>2</sub> 1260 VBr<sub>3</sub> 1260 VC 1288 V<sub>2</sub>C 1288 V(CH<sub>3</sub>COO)<sub>3</sub> 1283 V(C<sub>6</sub>H<sub>6</sub>)<sub>2</sub> 1289 VCl<sub>2</sub> 1255 VCl<sub>3</sub> 1256 VCl<sub>3</sub>.6H<sub>2</sub>O 1256 VCl<sub>4</sub> 1259 VF<sub>3</sub> 252 VF<sub>4</sub> 252 VF<sub>5</sub> 253 V-H 1296 VI<sub>2</sub> 1261 VI<sub>3</sub> 1262 VN 1286 V<sub>2</sub>N 1287 VO 1268 VO<sub>2</sub> 1267 V<sub>2</sub>O<sub>3</sub> 1267 V<sub>2</sub>O<sub>5</sub> 1270 V<sub>6</sub>O<sub>13</sub> 1266  $V_n O_{2n-1}$  1267 VOCI 1262 VOCl<sub>2</sub> 1263 VOC13 1264 VO<sub>2</sub>Cl 1265 V(OH)<sub>3</sub> 1268 VOSO<sub>4</sub> 1285 VOSO4·3H2O 1285  $VP_{<1}$  1287 VP 1287 VP<sub>2</sub> 1287 VS 1274 VS<sub>4</sub> 1275  $V_2S_3$  1275 V<sub>2</sub>(SO<sub>4</sub>)<sub>3</sub> 1283 VSO<sub>4</sub>·6H<sub>2</sub>O 1277 VSi<sub>2</sub> 1792

# W

W 1417,1622 W(CO)<sub>6</sub> 1741 WCl<sub>5</sub> 1419 WCl<sub>6</sub> 1420 

# Y

YF<sub>3</sub> 246 YbBr<sub>2</sub> 1150 YbCl<sub>2</sub> 1150 YbI<sub>2</sub> 1150 YbSO<sub>4</sub> 1138

# Z

Zn 1067 ZnAs<sub>2</sub> 1083 Zn<sub>3</sub>As<sub>2</sub> 594, 1083 ZnBr<sub>2</sub> 1072  $Zn(CH_3COO)_2$  1087  $Zn(C_2H_5)_2$  1084 Zn(CN)<sub>2</sub> 1087 ZnCO<sub>3</sub> 1086 ZnCl<sub>2</sub> 1070 ZnF<sub>2</sub> 242 ZnFe<sub>2</sub>O<sub>4</sub> 1090 ZnH<sub>2</sub> 1069 ZnI<sub>2</sub> 1073 Zn<sub>3</sub>N<sub>2</sub> 1080  $Zn(NH_2)_2$  1079 [Zn(NH<sub>3</sub>)<sub>4</sub>]MoO<sub>8</sub> 1414 ZnO 1664  $Zn(OH)_2$  1074 Zn(OH)Cl 1071 [Zn(OH)3]Na 1681  $[Zn(OH)_4]Na_2$  1682 Zn<sub>2</sub>(OH)PO<sub>4</sub> 1082 5ZnO-12WO3-aq 1713

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2			
Zn <sub>3</sub> P <sub>2</sub> 1080	Zr 1172	ZrOCl <sub>2</sub> .8H <sub>2</sub> O 1210	
Zn <sub>3</sub> (PO <sub>4</sub> ) <sub>2</sub> .4H <sub>2</sub> O 1081	ZrBr <sub>4</sub> 1203	$ZrO(NO_3)_2$ 1241	
ZnS 1075	ZrC 1245	ZrP 1241	
ZnS <sub>2</sub> O <sub>4</sub> 394	ZrCl <sub>4</sub> 1203	ZrP <sub>2</sub> 1241	
$Zn(SO_2 \cdot CH_2OH)_2$ 1076	$ZrF_4$ 251	ZrS <sub>2</sub> 1226	
$Zn_3(SbS_4)_2$ 1083	Zr-H 1185	ZrS <sub>3</sub> 1226	
ZnSe 1078	ZrI <sub>4</sub> 1205	Zr(SO <sub>4</sub> ) <sub>2</sub> 1231	
ZnSiF <sub>6</sub> .6H <sub>2</sub> O 1090	ZrN 1233	Zr(SO <sub>4</sub> ) <sub>2</sub> .4H <sub>2</sub> O 1231	
Zn <sub>2</sub> SiO <sub>4</sub> 1089	ZrO <sub>2</sub> 1220	ZrSi <sub>2</sub> 1249, 1792	

# Subject Index

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2

This index is arranged alphabetically in terms of the major element, group or unit constituting the compound. Numerical prefixes such as mono, di, tri, etc., as well as poly are disregarded for purposes of arrangement (except that, for example, hexaphosphates follow pentaphosphates). The same applies to ortho, meta, para and pyro but not to per, sub, hypo and oxo.

Prefixes occurring at the beginning of the first word are not underlined. They are underlined within the word to help distinguish the key roots.

# Α

Acetates-see parent cation Acetylacetonate, aluminum 836 chromium 1383 Active metals 1613 Active metal oxides 1656  $\alpha$ -Alaninate, chromium 1383 Alkali aluminates 1692 Alkali ammine graphite compounds 637 Alkali graphite compounds 635 Alkali hydrides 971 Alkali metal antimonides 985 arsenides 985 bismuthides 985 carbides 987 carbonates 987 germanides 989 niobates 1323 oxides 974 dioxides 980 phosphides 985 silicides 989 tantalates 1323 uranates (V) 1445 uranates (VI) 1445 Alkali metals, free 956 Alkyl ester of silicic acid 702 Alum, ammonium gallium (III) 854 ammonium vanadium 1284 cesium 956 cesium manganese 1468 cesium vanadium (III) 1284 potassium vanadium (III) 1284 rubidium vanadium (III) 1284 Aluminate, ammonium tetrafluoro-227 hexafluoro-226 cobalt 1525 monopotassium oxohydroxo- 1693

Aluminate, sodium tetrachloro-816 monosodium oxohydroxo-(I) 1693 oxohydroxo-(II) 1693 tetrasodium heptahydroxo- 1692 Aluminum acetate 835 acetylacetonate 836 amalgam 1807 arsenide 831 azide 476, 829 boride 772 bromide 813 bromide-hydrogen sulfide adduct 819 calcium hydride 806 carbide 832 chloride 812 chloride ammoniate 817 chloride graphite 644 chloride hydrate 815 chloride-phosphorus pentachloride adduct 818 chloride-sulfur dioxide adduct 817 chloride-thionyl chloride adduct 818 tetrachloroaluminic acid dietherate 816 chlorohydride 808 triethanolaminate 835 triethyl- 810 diethyl-, bromide 809 triethyl-, etherate 811 diethyl-, hydride 811 ethoxide 834 fluoride 225 hydride trimethylaminate 809 hydride, polymeric 807 hydroxide 820 hydroxide gel 1652 iodide 814 iodide-hexaammoniate 819 lithium cyanide 833 lithium hydride 805

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Aluminum lithium nitride 828 lithium phosphide 830 methoxide 833 nitride 827 oxide 822, 1660 orthophosphate 831 phosphide 829 selenide 825 sulfide 823 basic 824 neutral 824 telluride 826 Amalgams 1801 1807 Amalgam, aluminum barium 1805, 1806 bismuth 1806 cadmium 1806 calcium 1804 lead 1806 potassium 1803 rare earth 1807 sodium 1802 strontium 1805 tin 1806 zinc 1806 Amide, lithium 463 sodium 465 tetraAmide, pyrophosphoryl 588 triAmide, phosphoryl 584 thiophosphoryl 587 monoAmidophosphate, <u>di</u>sodium 581 diAmidophosphoric acid 582 monoAmidophosphoric acid 579 Amidosulfonate, potassium 507 Amidosulfonic acid 508 Aminate, borine trimethyl- 778 trans-diAmminedichloroplatinum (ll) 1578 hexaAmminechromium (III) chloride 1351 nitrate 1351 triAmminechromium tetroxide 1392 hexaAmminecobalt (II) chloride 1516 (III) chloride 1516 (III) nitrate 1526 Ammine complexes of platinum (II) 1577 decaAmmine-µ-peroxocobalt (III) cobalt (IV) sulfate 1540 tetraAmminecopper (II) sulfate 1021 diAmminemercury (II) dichloride 1114 hexaAmminenickel (II) chloride 1545 diAmminepalladium (II) salts 1585 tetraAmminezinc tetraperoxomolybdate (VI) 1414 Ammonia 460 from labeled NH4Cl 461 solutions of 463 Ammoniate, boron trifluoride 785 Ammoniated mercuric chloride 1114 Ammonium aluminum fluoride 226 orthoarsenate 602 hexabromoantimonate 615 carnallite 908, 951

Ammonium chlorate 313 hexachloroiridate (IV) 1594 -osmate (IV) 1603 tetrachloropalladate (II) 1584 chloroplatinate 1570 hexachloroplatinate (IV) 1570 -plumbate 751 -rhodate (III) 1588 -ruthenate (IV) 1599 -selenate 425 731 -stannate -tellurate 444 -titanate 1199 tetrachlorozincate 1072 cryolite 226 hexacyanoferrate (11) 1509 fluoride 183 tetrafluoroaluminate 227 hexafluoroaluminate 226 tetrafluoroberyllate 232 hexafluorogallate 228 hexafluoroindate 229 difluorophosphate 196 hexafluorophosphate 195 gallium alum 854 gallium (III) sulfate 854 hydrogen fluoride 183 hydrogen sulfide 357 iodide 289 iridium (IV) chloride 1594 lead (IV) chloride 751 1723 luteophosphotungstate hexanitrorhodate (III) 1586 heptanitrosyltrithioferrate 1764 1413 oxopentachloromolybdate (V) palladium (IV) chloride 1584 palladochloride 1584 pentaperoxodichromate 1392 peroxydisulfate 390 perrhenate 1484 persulfate 390 platinichloride 1570 plumbic chloride 751 rhodanilate 1378 ruthenium (IV) chloride 1599 salt of rhodanilic acid 1378 diAmmonium pentasulfane 369 Ammonium disulfatovanadate (III) 1283 pentasulfide 369 thioarsenate 604 trithiocarbonate 674 tetrathiocyanatodiamminechromate (III) 1376 -dianilinochromate (III) 1378 tetrathiomolybdate 1416 paratungstate 1713 uranyl carbonate 1449 metavanadate 1272 vanadium (III) alum 1284 zinc sulfate 1077

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Anhydrous sodium tetraborate 793 Antimonate, ammonium hexabromo- 615 sodium thio- 619 Antimonic acid, hexachloro- 611 Antimonide, gallium 857 indium 867 lithium 985 potassium 986 sodium 986 Antimony 606 explosive 1638 Antimony bromide (III) 613 chloride (III) 608 chloride (V) 610 dichloride trifluoride 200 fluoride (III) 199 fluoride (V) 200 hydrated oxide (V) 617 hydride (stibine) 606 iodide (III) 614 (III) oxide 615 (V) oxide 616 diAntimony tetroxide 618 Antimony (III) oxide chloride 611 oxide sulfate 619 (III) sulfate 618 Aquopentaamminecobalt (III) oxalate 1533 diAquotetraamminecobalt (III) sulfate 1537 hexaAquochromium (III) acetate 1371 (III) chloride 1348 orthoArgentite, sodium 1039 Argentous sulfide 1039 Arsenate, ammonium ortho- 602 boron 797 hydroxylammonium 501 sodium dihydrogen ortho- 602 triArsenate, sodium hydrogen 1709 Arsenates, isopoly- 1709 1-Arsenates, <u>12</u>-molybdic acid- 1734 12-tungstic acid- 1724 2-Arsenates, 6-molybdic acid- 1736 18 molybdic acid- 1734 18-tungstic acid- 1725 Arsenic 591 yellow 592 orthoArsenic acid 601 Arsenic tribromide 597 trichloride 596 fluoride (III) 197 fluoride (V) 198 diiodide 598 triiodide 597 diArsenic pentasulfide 603 trioxide 600 tetraArsenic tetrasulfide 603 Arsenide, gallium 857 indium 867 lithium 985 magnesium 917 potassium 986 sodium 986

Arsenide, sodium dihydrogen 595 Arsine 593 Aurate (III), hydrogen tetrachloro- 1057 Aurate, potassium 1061 potassium tetrachloro-1058 Aurate (1), potassium dicyano-1065 Auric chloride 1056 hydroxide 1060 oxide 1059 potassium chloride 1058 Aurochlorohydric acid 1057 Aurous chloride 1055 s-triAzaborane 779 Azide 474, 959 aluminum 829 barium 942 beryllium 899 lead 763 magnesium 917 Azodisulfonate, potassium 510

#### В

Barium 922, 928 amalgam 1805, 1806 azide 942 930 bromide chlorate 314 chloride 930 chromate 316 chromate (V) 1394 orthochromate (IV) 1393 tetracyanoplatinate (11) 1576 fluoride 234 hydride 929 dihydrogen hypophosphite 562 hexahydroxocuprate (II0 1685 hexahydroxoferrate (111) 1690 heptahydroxoferrate (III) 1690 hypophosphite 557 iodide 930 manganate (VII) 1462 nitride 941 oxohydroxostannate (II) 1696 oxide 933 perchlorate 320 periodate 326 permanganate 1462 peroxide 937 peroxide octahydrate 937 perrhenate 1487 1487 mesoperrhenate platinocyanide 1576 rhenate (VI) 1485 purification of salts of 933 selenide 939 disilicate 706 metasilicate 706 silicide 947 sulfate-potassium permanganate solid solution 1463

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Barium sulfide 938 telluride 940 trithiocarbonate 674 dithionate 397 dithiophosphate 571 Bayerite 821 diBenzenechromium (0) 1395 (I) iodide 1397 diBenzenemolybdenum (0) 1402 diBenzenevanadium (0) 1289 (Berthollet's) fulminating or detonating 1046 silver 232 Beryllate, ammonium tetrafluoro-895 sodium Beryllium 887 acetate 901 acetate, basic 901 azide 899 bromide 891 carbide 899 carbonate 893 chloride 889 formate, basic 902 fluoride 231 hydroxide 894 iodide 892 nitride 898 oxide 893 propionate, basic 902 selenide 897 sulfide 895 telluride 897 Bismuth 620 amalgam 1806 borate 627 (III) bromide 623 chloride 621 dichloride 622 fluoride 201 (V) fluoride 202 (III) iodide 624 diBismuth tetroxide 629 Bismuth oxide bromide 624 oxide chloride 622 oxide iodide 625 oxide nitrite 626 (III) phosphate 626 Bismuthate, potassium 628 sodium 627 anhydrous 627 hydrous 628 Bismuthide, 1ithium 985 potassium 986 sodium 986 Bisulfide, carbon 652 Boehmite 821 diBorane 773 Borate, bismuth 627 lithium aluminum 796 nitrosyl fluoro- 224 potassium fluoro- 223

Borate, potassium hydroxyfluoro-224 metaBorate, sodium 791 orthoBorate, sodium 790 tetraBorate, sodium 793 pentaBorate, sodium 795 Borate, sodium fluoro- 222 sodium tetraphenyl- 803 1-Borates, 12-tungstic acid-1716 Borax 793 Borazole 779 s-trichloro- 779 metaBoric acid 791 Borides 1798 aluminum 772 Borine trimethylaminate 778 Borofluoride, sodium 222 Borohydride, lithium 775 sodium 776 sodium trimethoxy 777 Boron 770 tetragonal 772 Boron trialkyls 800 arsenate 797 azide 476 tribromide 781 trichloride 780 triethyl- 799 trifluoride 219 trifluoride ammoniate 785 etherate 786 difluoride, n-butyl- 802 trifluoride dihydrate 784 triiodide 782 methoxide 797 trimethyl-798 nitride 789 oxide 787 phosphate 796 (III) sulfide 788 Boronic acid, n-butyl- 801 Boroxine, trimethyl- 800 tri-n-butyl- 801 monoBromamine 480 Bromate, barium 316 potassium <u>tetraflu</u>oro (III)-237 Bromic acid 315 Bromide—See parent cation Bromine 275 purification of 275 Bromine trifluoride 156 pentafluoride 158 graphite 643 hydrate 276 (III) nitrate 328 <u>di</u>oxide 306 diBromine monoxide 307 hexaBromoantimonate, ammonium 615 Bromofluorophosgene 210 diBromoiodide, cesium 297 potassium 296 triBromotripyridinemolybdenum 1408

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 triBromodisulfane 377 -trisulfane 379 -terrasulfane 379 -pentasulfane 379 -hexasulfane 379 -heptasulfane 379 -heptasulfane 379 -beptasulfane 379 -beptasulfane 379 -betasulfane 379 -betasulfane 379 tetraBromothallate (III), rubidium 876 thallium (I) 875

# С

Cadmate, potassium tetracyano- 1106 Cadmium (needles) 1092 Cadmium acetate 1105 amalgam 1806 amide 1100 arsenide 1103 bromide 1096 carbonate 1104 chloride 1093 chloride hydroxide 1094 cyanide 1105 diethyl- 1103 ferrate (III) 1107 fluoride 243 hvdroxide 1097 hydroxychloride 1094 iodide 1096 nitride 1100 phosphides 1101 potassium chloride 1095 cyanide 1106 rhodanide 1106 selenide 1099 silicate 1107 orthosilicate 1107 sulfide 1098 thiocyanate 1106 Calcium 922 aluminum hydride 806 amalgam 1801 bromide 930 carbide 943 chloride 930 cyanamide 946 fluoride 233 germanide 948 hydride 929 hydroxide 934 iodide 930 nitride 940 oxide 931 perchlorate 320 peroxide 936 peroxide hydrate 937 phosphide 527,942

Calcium orthoplumbate 760 Calcium salts, purification of 931 selenide 939 silicide 946 sulfide 938 telluride 940 832 Carbide, aluminum beryllium 899 calcium 943 lithium 987 magnesium 920 sodium 987 Carbon 630 adsorptive (activated) 633 black, graphitized 631 lustrous 632 monoxide black 631 oxygen compounds reacting as acids 634 oxygen compounds reacting as bases 634 preparations, special 631 sulfur compounds 634 surface compounds of 633 Carbon monofluoride 640 tetrafluoride 203 tetraCarbon monofluoride 641 Carbon monoxide 645 purification of tank CO 646 Carbon dioxide 647 purification of tank CO<sub>2</sub> 647 triCarbon dioxide 648 Carbon selenide 656 diselenide 656 suboxide 648 subsulfide 653 disulfide 652 triCarbon disulfide 653 Carbonate-see under parent cation Carbonatotetraamminecobalt (III) sulfate 1535 Carbonyl bromofluoride 210 chloride 650 chlorofluoride 209 fluoride 206 difluoride 206 iodofluoride 211 iron 1636 selenide 655 sulfide 654 Caro's acid 389 Cementite 1503 Cerium metal, solid 1142, 1145 Cerium monochalcogenide 1155 compounds, pure 1132 earths, purification of 1131 (III) fluoride 247 (IV) fluoride 247 (III) oxide 1151 Cesium 957 alum 956 azide 959

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Cesium dibromoiodide 297 carbonate 988 chloride 961 from carnallite 951 from dollucite 955 <u>di</u>chlorobromide 294 dichloroiodide 296 hexachloroselenate 425 hexachlorotellurate 444 nonachlorodithallate (III) 874 chromate 1389 dichromate 1389 dioxide 981 germanide 989 graphite 635 hydride 971 hydroxide 983 manganese alum 1468 manganese (III) sulfate 1468 oxide 978 silicide 989 vanadium (III) alum 1284 Cesium, purification by vacuum distillation 963 Chalcogenides, europium (II) 1155 monoChalcogenides, cerium 1155 lanthanum 1155 neodymium 1155 praseodymium 1155 Chinese red 1112 monoChloramine 477 Chlorate-see parent cation Chloric acid 312 Chlorides-see parent cation Chlorine 272 azide 476 monofluoride <u>tri</u>fluoride 155 hvdrate 274 nitrate 326 dioxide 301 diChlorine oxide 299 hexoxide 303 heptoxide 304 Chlorine dioxide fluoride 165 trioxide fluoride 166 tetroxide fluoride 167 Chlorite, sodium 312 tetraChloroaluminate, sodium 816 tetraChloroaluminic acid <u>di</u>etherate 816 triChlorotriamminechromium 1381 -pyridinechromium 1381 Chloropentaamminecobalt (III) chloride 1532 -chromium (III) chloride 1352 diChlorotetraamminecobalt (III) chloride 1536 1,2-diChlorotetraamminecobalt (III) chloride (cis) 1536 1,6-diChlorotetraamminecobalt (III) chloride (trans) 1537

Chloropentaamminerhodium salts 1590 chloride 1590 nitrate 1590 Chloroantimonate (V), nitrosyl 612 hexaChloroantimonic acid (V) 611 diChloroaquotriamminechromium (III) chloride 1358 triChlorotriaquochromium 1380 Chloropentaaquochromium (III) chloride 1350 sulfate 1350 pentaChloroaquothallate (III), potassium 874 tetraChloroaurate (III), potassium 1058 Chloroauric acid 1057 s-triChloroborazole 779 1390 Chlorochromate, potassium triChlorodialane 808 triChlorotriethanolochromium 1380 cis-diChlorodiethylenediaminechromium (III) chloride 1356 trans-diChlorodiethylenediaminechromium (III) chloride 1357 205 diChlorotetrafluoroethane triChloro<u>di</u>fluoroethane 205 -trifluoroethane 205 monoChlorotrifluoromethane 205 diChloro<u>di</u>fluoromethane 205 Chlorofluorophosgene 208 diChlorofluorophosphine 191 triChlorogermane 721 Chlorohydride, aluminum 808 508 Chloroimidosulfonate, potassium tetraChloroiodic acid 299 diChloroiodide, cesium 296 potassium 295 tetraChloroiodide, potassium 298 hexaChloroiridate (IV), ammonium 1594 (III), potassium 1595 (IV), potassium 1593 hexaChloroiridium (IV) acid 1593 -manganate (IV), potassium 1464 -molybdate (III), potassium 1408 -osmate (IV), ammonium 1603 sodium 1602 tetraChloropalladate (II), ammonium 1584 potassium 1584 sodium 1584 hexaChloropalladate (IV), ammonium 1584 potassium 1584 Chlorophosgene 209 tetraChlorophosphonium hexafluorophosphate (V) 193 Chloroplatinate (IV), ammonium 1570 tetraChloroplatinate (II), potassium 1572 hexaChloroplatinate (IV), ammonium 1570 potassium 1571 sodium 1571 Chloroplatinic acid 1569 tetraChloroplatinic (II) acid 1570 hexaChloroplatinic (IV) acid 1569

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 hexaChloroplumbate, ammonium 751 potassium 753 Chloropraseocobalt chloride 1537 Chloropurpureocobalt chloride 1533 triChlorotripyridinechromium 1381 hexaChlorotripyridineditungstate (III) 1429 hexaChlororhodate (III), ammonium 1588 potassium 1588 sodium 1587 hexaChlororuthenate (IV), ammonium 1599 hexaChloroselenates 425 ammonium 425 cesium 425 potassium 425 rubidium 425 thallium 425 Chlorosilane 691 diChlorosilane, dimethyl 694 Chlorosiloxanes 695 hexaChlorostannate, ammonium 731 potassium 731 hexaChlorostannic acid 730 diChloromonosulfane 370 -disulfane 371 -trisulfane 372 -tetrasulfane 372 -pentasulfane 372 372 -hexasulfane -heptasulfane 372 -octasulfane 372 Chlorosulfonic acid 385 hexaChlorotellurates 444 ammonium 444 cesium 444 potassium 444 rubidium 444 thallium 444 tetraChlorothallate (III), thallium (1) 872 hexaChlorothallate (III), potassium 873 thallium (1) 873 nonaChlorodithallate (III), cesium 874 tetraChlorothallium (III) acid 872 enneaChloroditungstate (III), potassium 1427 Chlorovioleocobalt chloride 1537 Chloryl fluoride 165 oxyfluoride 166 diChlorylphosphoric acid anhydride 536 Chromate, cesium 1389 potassium chloro-1390 potassium fluoro- 1390 rubidium 1388 Chromate (11), potassium hexathiocyanato-1374 Chromate (III), ammonium tetrathiocyanatodianilino-1378 ammonium tetracyanatodiammine-1376 trilithium hexaphenyl- 1375 potassium trioxalato- 1372 potassium tetrathiocyanatodipyridino-1379

Chromate (III), sodium <u>hexa</u>hydroxo- 1688 Chromate (IV), barium ortho- 1393 Chromate (V), barium 1394 potassium tetraperoxo-1391 diChromate, ammonium pentaperoxo- 1392 cesium 1389 rubidium 1388 triChromate, potassium 1709 tetraChromate, potassium 1710 Chromates, isopoly- 1709 Chromic (III) acid, tetrathiocyanatodiammine- 1377 chloride, luteo- 1351 nitrate, luteo- 1351 -2-Chromites, 12-molybdic acid- 1737 Chromite, sodium thio- 1394 Chromium 1334 Chromium (II) acetate 1368 (III) acetate, <u>hexa</u>aquo- 1371 triChromium (III) acetate, <u>di</u>hydroxohexaacetato-1371 Chromium (III) acetylacetonate 1383 (0) dibenzene- 1395 (II) bromide 1340 (III) bromide 1341 carbonyl 1741 tricarbonyl, tripyridine 1749 tetracarbonyl, dipyridine 1749 (II) chloride 1336 (III) chloride 1338 hexaammine- 1351 <u>hexa</u>aquo- 1348 chloropentaammine- 1352 chloropentaaquo- 1350 dichloroaquotriammine- 1358 chloride, erythro- 1360 (III) chloride, cis-dichlorodiethylenediamine- 1356 trans-dichlorodiethylenediamine- 1357 <u>tri</u>ethylenediamine 1354 triChromium (III) chloride, dihydroxohexaacetato- 1371 Chromium (III) chloride, hexaurea- 1359 trichlorotriammine-1381 -triaquo- 1380 -triethanolo-1380 -tripyridino-1381 (II) fluoride 256 257 (lll) fluoride (IV) fluoride 258 (III) glycinate 1382 (III) hydroxide 1345 (II) iodide 1344 (III) iodide 1344 (I) iodide, dibenzene-1397 bis(diphenyl)- 1397 (diphenyl)(benzene)-1398 (III) nitrate, hexaammine- 1351 nitride 1347 (II) oxalate 1370 <u>tetr</u>oxide triammine 1392

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Chromium (III) oxide, copper 1672 oxide gel, hydrated 1648 trioxide-pyridine 1385 perchlorate, tris(2,2 -dipyridyl)-1362 (II) perchlorate, <u>tris(2,2 dipyridy1)</u>- 1361 (IV) diperoxotriammine-1392 (0) <u>bis(diphenyl)</u>- 1396 (0) hexaphenylisonitrilo-1363 orthophosphate 1364 (0) <u>tris</u> (2,2'-<u>dipyridyl</u>)-1363 (II) sulfate 1363 (III) sulfate, <u>tri</u>ethylenediamine 1354 sulfides 1346 (III) thiocyanate, triethylenediamine-1354 trans-dithiocyanatodi(ethylenediamine)-1357 (III) xanthate 1383 Chromium (II) salt solutions 1366 Chromyl chloride 1384 fluoride 258 nitrate 1384 perchlorate 1387 Cinnabar 1112 Cinnabar green 1092 Cinnabarite 1112 Clerici's solution 884 Cobalt, metallic 1513 pyrophoric 1615 very pure 1513 Cobalt aluminate 1525 (III) amide 1526 (II) bromide 1517 (III) bromide, <u>tri</u>ethylenediamine- 1538 bromide, dinitrosyl 1761 (III) bromide tartrate, d-triethylenediamine- 1539 diCobalt carbide 1531 Cobalt carbonyls 1746 tetracarbonyl hydride 1753 carbonyl mercury 1755 (II) chloride 1515 hexaammine 1516 (III) chloride, <u>hexaammine-</u> 1531 chloropentaammine- 1532 1,2-dichlorotetraammine-1536 1,6-dichlorotetraammine-1537 nitritopentaammine- 1535 nitropentaammine- 1534 chloride, dinitrosyl 1761 (III) hexacyanide, potassium 1541 (II) fluoride 267 (III) fluoride 268 green 1092 (II) hydroxide 1521 (III) hydroxide 1520 (II) iodide 1518 iodide, dinitrosyl 1761 (III) nitrate, hexaammine-1526 nitride 1529 diCobalt nitride 1529

Cobalt nitrosyl tricarbonyl 1761 dinitrosyl halides 1761 (III) oxalate, aquopentaammine-1533 (II) oxide 1519 (II, III) oxide 1520 (III) oxide 1675 phosphide 1530 (III) sulfate 1524 (IV) sulfate, <u>deca</u>ammine-µ-peroxocobalt (III) 1540 (III) sulfate, <u>diaquotetraammine</u> 1537 (III) sulfate, carbonatotetraammine- 1535 sulfides 1523 sulfoxylate 393 Cobaltate (III), potassium hexacyano-1541 (IV), potassium heptafluoro- 269 (III) sodium hexanitrito- 1541 Cobaltic (III) acid, hexacyano- 1542 Copper, pure metal 1003 active 1633 colloidal 1003 Copper acetoarsenite 1027 acetylide 1026 (II) azide 1022 (I) bromide 1006 (II) bromide 1009 carbonates (basic) 1024 (I) chloride 1005 (II) chloride 1008 chromium oxide 1672 (II) fluoride 238 hydride 1004 (II) hydroxide 1013 (I) iodide 1007 (I) tetraiodomercurate (II) 1110 (I) nitride 1021 (I) oxide 1011 (II) oxide 1012 oxychloride 1010 phosphide 1023 diphosphide 1024 (I) selenide 1019 (I) sulfate 1020 (II) sulfate, tetraammine- 1021 (I) sulfide 1016 (II) sulfide 1017 (I) telluride 1019 Corundum 822 Cuprate (III), potassium hexafluoro- 269 Cupric bromide 1009 carbonate, blue 1025 chloride 1008 hydroxide 1013 oxide 1012 sulfide 1017 Cuprous azide 1022 bromide 1006 chloride 1005 iodide 1007 nitride 1021 oxide 1011

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Cuprous selenide 1019 sulfate 1020 sulfide 1016 telluride 1019 Cyanamide, calcium 946 silver 1047 Cyanate, silicon 702 Cyanic acid 667 Cyanide, hydrogen 658 lithium aluminum 833 Cyanoamminoferrate (II), sodium penta-1511 (III), sodium penta- 1512 Cyanoaurate (I), potassium di-1065 tetraCyanocadmate, potassium 1106 Cyanochromate (III), potassium hexa-1373 Cyanocobaltate (III), potassium hexa-1541 hexaCyanocobaltic (III) acid 1542 hexaCyanoferrate (II), ammonium 1509 sodium carbonyl 1769 sodium nitrosyl 1768 hexaCyanoferric acid, (II) 1509 (111) 1510 Cyanogen 660 diCyanogen 660 Cyanogen bromide 665 chloride 662 iodide 666 Cyanomanganate (I), potassium <u>hexa</u>-1472 (II), potassium <u>hexa</u>- 1473 (III), potassium hexa- 1474 potassium nitrosyl 1767 Cyanomercurate (II), potassium tetra 1122 Cyanomolybdate (IV), potassium octa-1416 potassium nitrosyl 1766 Cyanonickelate (II), potassium tetra-1559 Cyanoplatinate (II), barium tetra-1576 potassium tetra- 1576 Cyanotungstate (IV), potassium octa- 1429 (V), potassium octa-1430 Cyanozincate, potassium tetra-1088 Cyanuric acid 668

# D

Deuteride, hydrogen 126 Deuterium 121 Deuterium bromide 131 chloride 129 fluoride 127 iodide 133 sulfide 134 Deuteroammonia 137 -phosphoric acid 138 -sulfuric acid 135 (Dipheny1)(benzene)chromium (I) iodide 1398

# Е

Erythrochromium chloride 1360 triEthanolaminate, aluminum 835 Etherate, boron trifluoride 786 Ethiop's mineral 1111 Ethoxide, aluminum 834 tetraEthoxygermane 725 -silane 702 triEthylaluminum 810 diEthylaluminum bromide 809 triEthylaluminum etherate 811 diEthylaluminum hydride 811 diEthylcadmium 1103 triEthylenediaminecobalt (III) bromide 1538 optical antipodes of 1538 triEthylenediaminechromium (III) chloride 1354 sulfate 1354 thiocyanate 1354 Ethylenediamine iron carbonyl 1756 triEthylenediamine iron (II) octacarbonyl diferrate (11) 1756 diEthylmercury 1118 Ethyl dinitrosyl thioferrate 1765 diEthylzinc 1084 Europium compounds, pure 1136 Europium (II) bromide 1150 (II) carbonate 1137 (II) chloride 1136 (II) fluoride 248 (II) iodide 1150 (II) oxide 1156 (III) oxide 1156 (II) selenide 1155 (II) sulfate 1137 (II) sulfide 1155 (II) telluride 1155

# F

Fehling's solution 1027 Ferrate (II), ammonium hexacyano- 1509 Ferrate, ammonium heptanitrosyl trithio-1764 <u>tetra</u>-Ferrate (III), barium hexahydroxo-1691 (III), barium heptahydroxo-1691 (III), 1ithium 1504 (VI), potassium 1504 (IV), potassium hexafluoro- 269 potassium nitrosyl <u>tri</u>carbonyl 1759 potassium dinitrosyl thiosulfato- 1766 sodium carbonyl cyano-1769 (II), sodium pentacyanoammino 1511 (III), sodium pentacyanoammino 1512 (II), sodium pentacyanoaquo- 1769 (II), sodium <u>tetra</u>hydroxo- 1686 (III), sodium heptahydroxo-1689 (III), sodium octahydroxo- 1690

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Ferrate (III), sodium nitrosyl cyano-1768 (III), sodium hexathiocyano-1511 Ferric (II) acid, hexacyano- 1509 (III) acid, <u>hexa</u>cyano-1510 Ferric bromide 1494 chloride 1492 Ferricyanic acid 1510 Ferrocyanic acid 1509 1493 Ferrous bromide chloride 1491 iodide 1495 oxide 1497 sulfide 1502 Fluorides-see parent cations Fluorine 143 Fluorine nitrate 187 hexaFluoroaluminate, ammonium 226 tetraFluoroaluminate, ammonium 227 tetraFluoroberyllate, ammonium 232 Fluoroborate, nitrosyl 224 223 potassium potassium hydroxy- 223 sodium 222 Fluorochromate, potassium 1390 Fluoroboric acid 221 Fluoroboric acid, <u>di</u>hydroxy- 784 tetraFluorobromate (III), potassium 237 Fluoroform 204 Fluoroformyl fluoride 206 hexaFluorogallate, ammonium 228 -germanate, potassium 216 -indate, ammonium 229 heptaFluorocobaltate (IV), potassium 269 hexaFluoroiodate (V), potassium 238 tri Fluoroiodomethane 205 hexaFluoromanganate (IV), potassium 264 tri Fluoromethane 204 hexaFluoronickelate (IV), potassium 269 255 heptaFluoroniobate (V), potassium diFluorophosphate (V), ammonium 196 hexaFluorophosphate (V), ammonium 195 tetrachlorophosphonium 193 196 potassium triFluorosilane 214 tetraFluorosilane 212 Fluorosilicate, zinc 1090 Fluorosilicic acid 214 hexaFluorosilicic acid 214 Fluorosulfinate, potassium 178 Fluorosulfonic acid 177 heptaFluorotantalate, potassium 256 pentaFluorothorate, potassium 1177 hexaFluorotitanate, potassium 1163 sodium 1163 Fluosilicic acid 214 Formyl nitrile 658 Free alkali metals 956 Fremys salt 504

# G

Gadolinite, treatment of 1129 diGallane 840 tetramethyl 840 Gallate, ammonium hexafluoro-228 lithium tetrahydro- 842 837 Gallium dissolving in acids 839 electrolytic separation of 837 Gallium alum, ammonium 854 antimonide 857 arsenide 857 azide 476 (II) bromide 847 (III) bromide 845 (II) chloride 846 (III) chloride 843 (III) fluoride 227 hydroxide 847 metahydroxide 847 (III) iodide 846 840 trimethyl-856 nitrate nitride 855 (I) oxide 849 (III) oxide 848 (III) perchlorate 839 phosphide 857 selenide 854 (III) sulfate, ammonium 854 (1) sulfide 852 (II) sulfide 851 (III) sulfide 850 telluride 855 Gel, aluminum hydroxide 1652 hydrated chromium oxide 1648 hydrated oxide 1646 silica 1648 Germanate, potassium <u>hexa</u>fluoro-216 Germane, trichloro-721 tetraethoxy- 725 Germanide, calcium 948 989 cesium magnesium 922 potassium 989 rubidium 989 sodium 989 Germanium 706 metallic 712 Germanium tetraacetate 726 (IV) bromide 718 715 (IV) chloride <u>di</u>chloride 716 215 tetrafluoride hydrides 713 (IV) iodide 719 diiodide 720 triiodide, methyl-722 nitride 722 (II) oxide 711

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Germanium (IV) oxide 706 purification of 709 via acid decomposition 706 via alkaline decomposition 708 disulfide 723 monosulfide 724 Glycinate, chromium 1382 Gold, very pure 1052 colloidal 1053 from residues 1054 (I) acetylide 1063 (I) chloride 1055 (III) chloride 1056 monochloride 1055 trichloride 1056 trichloride acid 1057  $\overline{(1)}$  cyanide 933 (III) hydroxide 1060 (III) oxide 1059 sesquioxide 1059 trioxide 1059 sulfide 1061 (II) sulfide 1062 (III) sulfide 1063 Gold potassium cyanide 1065 See also aurous, auric, etc. Graham's salt 550 Graphite, alkali compounds 635 See also carbon Graphite, alkali ammine compounds 637 aluminum chloride 644 bisulfate 642 bromine 643 cesium 635 fluoride 640 foils and films 632 iron (III) chloride 644 metal halides 644 nitrate 643 oxide 638 oxide black 631 oxyhydroxide 638 perchlorate 643 potassium 637 rubidium 637 salts 642 sodium 637 Graphitic acid 638

#### Н

Hafnium 1172 separation from zirconium 1179 (IV) bromide 1203 carbide 1245 (IV) chloride 1203 nitride 1233, 1236 (IV) oxide 1221 oxychloride 1213 phosphate 1244 Halides, poly- 293

Heavy hydrogen and compounds-see Deuterium Heteropoly acids 1698 Heteropoly compounds 1699, 1716 Hittorf's phosphorus 520 Hopkalite (Hopcalite) 1674 Hydrargillite 820 Hydrate, bromine 276 chlorine 274 diHydrate, boron trifluoride 784 octaHydrates of alkaline earth metal peroxides 937 Hydrated chromium oxide gel 1648 Hydrated oxide gels 1646 Hydrazine 469 hydrate 469 disulfonate, potassium 509 Hydrazinium sulfate 468 Hydrazoic acid 472 Hydrides—see parent metal Hydrochloroauric acid 1057 Hydrocyanic acid 658 Hydrogen 111 aluminum chloride 815 bromide 282 chloride 280 tetrachloroaurate (III) 1057 hexacyanocobaltate (III) 1542 hexacyanoferrate (II) 1509 (III) 1510 deuteride 126 fluoride 145 fluoride, ammonium 183 potassium 237 diHydrogen hypophosphate, disodium 560 diHydrogen hypophosphite, barium 562 Hydrogen iodide 286 peroxide 140 diHydrogen phosphate, sodium 544 Hydrogen phosphide (gaseous) 525 diHydrogen phosphide, sodium 530 Hydrogen selenide 418 selenide, sodium 419 disulfatovanadate (III) 1282 sulfide, ammonium 357 sulfide, sodium 357 tetraHydrogentellurate, sodium 453 438 Hydrogen telluride thiocyanate 669 Hydroxides—see parent metal diHydroxohexaacetatotrichromium (III) acetate 1371 chloride 1371 hexaHydroxochromate (III), sodium 1688 tetraHydroxocuprate (II), sodium 1684 hexaHydroxocuprate (II), barium 1685 Hydroxoferrates (III), barium 1690 sodium 1689 tetraHydroxoferrate (II), sodium 1686 tetraHydroxomagnesate, sodium 1683 1686 hexaHydroxonickelate (ll), strontium hexaHydroxoplumbate, sodium 1694

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Hydroxo salts 1677 triHydroxostannate (II), sodium 1687 hexaHydroxostannate (IV), sodium 1694 Hydroxozincates, sodium 1681 Hydroxyapatite 545 Hydroxyfluoroborate, potassium 223 diHydroxyfluoroboric acid 784 Hydroxylamine 501 disulfonate, potassium 503 isomonosulfonic acid 510 Hydroxylammonium arsenate 501 chloride 498, 500 oxalate 501 phosphate 500 hexaHydroxyplatinates (IV) 1575 Hypersulfuric acid 388, 389 Hypobromite, potassium 311 sodium 310 Hypochlorite, sodium 309 Hypochlorous acid 308 Hyponitrates-see parent cation Hyponitrites—see parent cation Hyponitrous acid 492 Hypophosphate, <u>di</u>sodium <u>di</u>hydrogen 560 <u>tetra</u>sodium 561 Hypophosphite, barium 557 barium <u>di</u>hydrogen 557 Hypophosphoric acid 558 Hypophosphorous acid 555

#### l

lmide, lithium 464 heptasulfur 411 thionyl 480 Imidophosphate, tetrasodium 589 Imidosulfonate, potassium 506 potassium chloro- 508 Indate, ammonium hexafluoro-229 Indium 857 electrolytic refining of 858 prerefining of crude 857 antimonide 867 arsenide 867 (1) bromide 862 (II) bromide 861 (III) bromide 859 (1) chloride 862 (II) chloride 861 (III) chloride 858 (III) fluoride 228 hydroxide 862 (I) iodide 862 (II) iodide 861 (III) iodide 860 nitride 866 (I) oxide 863 (III) oxide 863 phosphide 867 selenides 865

sulfides 864 tellurides 865 Intermetallic compounds 1771 Iodate, iodine (III) 331 Iodic acid 316 tetrachloro-299 Iodine 277 recovery from laboratory waste solutions 278 lodine <u>mono</u>bromide 291 monochloride 290 trichloride 292 pentafluoride 159 heptafluoride 160 (III) iodate 331 (III) nitrate 329 (III) perchlorate 330 (III) sulfate 329 Iodine (I) perchlorate, dipyridine-327 dilodine tetroxide 333 pentoxide 307 lodine (III) sulfate, oxo-332 Iodofluorophosgene 211 lodohydrargyrate, potassium 1110 lodoplumbite, potassium 754 trilodomercurate (II), potassium 1110 tetralodomercurate (II), copper (I) 1110 silver 1110 Iridate (IV), ammonium hexachloro-1594 (III), potassium <u>hexa</u>chloro- 1595 (IV), potassium hexachloro- 1593 Iridium, pure 1590 (IV) acid, <u>hexachloro-</u> 1593 (III) chloride 1592 (IV) chloride, ammonium 1594 (III) chloride, potassium 1595 (VI) fluoride 270 (IV) oxide 1590 (III) oxide, hydrated 1592 (IV) oxide, hydrated 1591 1490 lron, metallic carbonyl 1636 Iron (III) acetate, basic 1508 trilron (III) hexaacetatodihydroxomonoacetate 1508 Iron (II) bromide 1493 (III) bromide 1494 carbide 1503 tetracarbonyl 1745 pentacarbonyl 1743 carbonyl, ethylenediamine 756 tetracarbonyl halides 1751 <u>di</u>hydride 1752 carbonyl mercury 1755 carbonyl, pyridine 1758 (II) chloride 1491 (III) chloride 1492 (III) chloride graphite 644 (II) fluoride 266 (III) fluoride 266 (II) hydroxide 1498

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Iron (III) hydroxide 1499 (II) iodide 1495 1502 nitrides dinitrosyl dicarbonyl 1760 dinitrosyl halides 1762 (II) oxide 1497 (IIÍ) oxide, α = 1661 (II, III) oxide 1499 (III) oxide, "glimmering" hydrated 1654 (III) oxychloride 1501 (III) sulfate, basic 1507 (II) sulfide 1502 (III) sulfide, potassium 1507 (III) thiocyanate, sodium 1511 dilron enneacarbonyl 1744 nonacarbonyl 1744 trilron dodecacarbonyl 1745 Isocyanate, silicon 702 Isopoly compounds 1702 arsenates 1709 chromates 1709 molybdates 1710 niobates 1705 sulfates 1714 tantalates 1707 tungstates 1712 vanadates 1702

## Κ

Kernite, synthetic 794 Kurrol's sodium polyphosphate 551

# L

Lanthanum metal, powder 1141 solid 1144 Lanthanum tribromide 1148 trichloride 1146 cyclopentadienides 1159 fluoride 246 hydroxide 1152 triiodides 1149 nitrate 1158 nitride 1157 nonaselenide 1155 selenides 1154 sulfate 1156 monosulfide 1155 sulfide 1153 monotelluride 1155 Lead 748 purest form 748 See also plumbate, plumbous, etc. Lead (IV) acetate 767 tetraacetate 767 amalgam 1806 azide 763

Lead carbonate 766 Lead chamber crystals 406 Lead (IV) chloride 750 tetrachloride 750 tetraethyl- 765 (II) fluoride 218 (IV) fluoride 219 tetramethyl- 763 (II, IV) oxide 755 (IV) oxide 757, 1668 dioxide 757 "superoxide" 757 red 755 metasilicate 705 (IV) sulfate 761 disulfate 761 sulfide 760 thiocyanate 769 Lepidolite 950 Lindemann glass 796 Lithium 956 aluminum cyanide 833 aluminum hydride 805 aluminum nitride 828 aluminum phosphide 830 amide 463 antimonide 985 arsenide 985 azide 475 beryllium borate 796 bismuthide 985 borohydride 775 carbide 987 carbonate 950, 987 chromate 958 ferrate (III) 1504 fluoride 235 hydride 971 tetrahydrogallate 842 hydroxide 982 imide 464 nitride 984 oxide 974 peroxide 979 triLithium hexaphenylchromate (III) 1375 Lithium phosphide 985 metasilicate 705 silicide 991 Lustrous carbon 632 Luteochromic chloride 1351 nitrate 1351 Luteocobalt chloride 1531 Luteophosphomolybdic acid 1732 Luteophosphotungstic acid 1724

#### Μ

Madrell's salt, sodium <u>poly</u>phosphate 549 Magnesate, sodium <u>tetra</u>hydroxo- 1683 Magnesium 903

#### 1840

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Magnesium arsenide 917 azide 917 bromide 909 carbides 920 carbonate 232 chloride 905 fluoride 232 922 germanide hydride 905 hydroxide 912 iodide 910 nickel mixed oxalate catalyst 1615 nitride 912 oxide 911, 1663 perchlorate 320 phosphide 917 selenide 915 silicide 921 sulfide 913 telluride 915 Magnus's salt 1577 Malachite 1025 Manganate (VII), barium 1462 (VI), potassium 1461 (IV), potassium hexachloro- 1464 (I), potassium hexacyano- 1472 (II), potassium <u>hexacyano</u> 1473 (III), potassium hexacyano- 1474 (IV), potassium hexafluoro- 264 potassium nitrosyl cyano- 1767 (III), potassium trioxalato- 1470 (IV), potassium dioxalatodihydroxy-1471 (VII), silver 1463 (V), sodium 1460 Manganese 1454 (III) acetate 1469 cesium alum 1468 (III) chloride, potassium 1464 (II) cyanide 1473 (III) cyanide 1474 (II) fluoride 262 (III) fluoride 263 hexafluoride 264 (II) hydroxide 1456 nitride 1468 (II) oxide 1455 (III) oxide 1457 (IV) oxide 1458 (VII) oxide 1459 dioxide 1458 heptoxide 1459 (III) sulfate 1467 (III) sulfate, cesium 1468 1465 (II) sulfide Mercurate (II), copper (I) tetraiodo-1110 (II), potassium <u>tetracyano</u>- 1122 (II), potassium <u>tri</u>iodo- 1110 (II), potassium tetrathiocyano- 1124 (II), silver tetraiodo-1111 Mercuric acetate 1120

Mercuric cyanide 1121 iminobromide 1115 oxychloride 1108 selenide 1113 sulfide, black 1111 sulfocyanate 1123 thiocyanate 1123 thionitrosylate 1118 Mercuric precipitate (infusible), white 1114 (fusible), white 1114 Mercurous acetate 1120 thiocyanate 1122 thionitrosylate 1117 Mercury 27 (I) acetate 1120 (II) acetate 1120 (II) amide chloride 1114 (II) bromide 1109 (II) dichloride, diammine- 1114 cobalt carbonyl 1755 cyanide 1121 diethy1- 1118 (I) fluoride 243 (II) fluoride 244 (II) iminobromide 1115 iron carbonyl 1755 (II) oxychloride 1108 potassium iodide 1110 (I) rhodanide 1122 (II) selenide 1113 (II) sulfide 1111 (I) thiocyanate 1122 (II) thiocyanate 1123 thionitrosylate 1117 (II) thionitrosylate 1118 Methoxide, aluminum 833 boron 797 tetraMethoxysilane 702 triMethylboroxine 800 diMethyldichlorosilane 694 tetraMethyldigallane 840 triMethylgallium 840 Methylgermanium triiodide 722 Millon's base 1116 bromide of 1117 Molybdate (VI), tetraamminezinc tetraperoxo- 1414 (V), ammonium oxopentachloro- 1413 ammonium tetrathio- 1416 (III), potassium hexachloro-1408 (IV), potassium octacyano- 1416 potassium hydrogen diperoxo mono-1414 potassium nitrosyl cyano-1766 Molybdates, isopoly 1710 ammonium 1711 sodium 1710, 1712 Molybdenum 1401 Molybdenum (0), dibenzene-1402 Molybdenum blue 1411

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Molybdenum (III) bromide 1407 tribromotripyridine- 1408 carbonyl 1741 (II) chloride 1403 (III) chloride 1404 (IV) chloride 1405 (VI) fluoride 259 hydroxides, lower 1411  $\gamma$ -oxide 1410 (IV) oxide 1409 (VI) oxide 1412 trioxide 1412 (IV) sulfide 1415 disulfide 1415 Molybdic acid 1412 6-Molybdic acid-2-arsenates 1736 12-Molybdic acid-1-arsenates 1734 18-Molybdic acid-2-arsenates 1734 12-Molybdic acid-2-chromites 1737 6-Molybdic acid-1-periodates 1738 12-Molybdic acid-1-phosphates 1730 18-Molybdic acid-2-phosphates 1732 12-Molybdic acid-1-silicates 1729 Molybdic anhydride 1412 Monazite sand, extraction of 1127 Mosaic gold 741

#### Ν

Neodymium monochalcogenides 1155 compounds from cerium earths 1131 Nickel 1543 (II) amide 1554 (II) bromide 1545 carbide 1556 (II) carbonate 1556 carbonyl 1747 dicarbonyl o-phenanthroline 1750 (II) chloride 1544 (II) chloride, <u>hexa</u>mmine- 1545 (II) cyanide hydrate, potassium 1559 (II) fluoride 269 formate-paraffin catalyst 1631 (II) hydroxide 1549 (II, III) hydroxide 1551 (III) hydroxide 1551 B- 1549 γ\_ 1550 (II) iodide 1547 -magnesium mixed oxalate catalyst 1615 hexaNitritocobaltate (III), sodium 1541 triNickel nitride 1555 dinitride 1555 Nickel (II) oxide 1548 (II) sulfide 1551 (IV) sulfide 1554 diNickel (IV), di-µ-sulfido-tetrakis (dithiobenzoato)- 1558 Nickel (II) thiocyanate 1558 Nickelate (II), potassium tetracyano-1559 (IV), potassium hexafluoro- 269

Nickelate (II), strontium hexahydroxo 1686 Niobates, alkali 1323 isopoly 1705 Niobate, potassium 1323 (V), potassium heptafluoro-255 potassium peroxy- 1325 sodium 1323 Niobic acid, peroxy-1324 Niobium 1292 (III) bromide 1309 (V) bromide 1311 carbides 1331 (II) chloride 1296 (III) chloride 1297 (IV) chloride 1299 (V) chlorides 1302 deuteride 1296 (V) fluoride 254 heptafluoride, potassium 255 hydride 1295 (11) iodide 1314 (III) iodide 1314 (IV) iodide 1314 (V) iodide 1315 nitrides 1328 (11) oxide 1317 (IV) oxide 1318 (V) oxide 1318 oxytribromide 1313 1307 oxy<u>tri</u>chloride phosphides 1330 sulfides 1327 Nitramide 496 Nitrates-see under parent cation Nitrates, hypo-see hyponitrates Nitric acid 491 anhydrous 491 Nitric oxide 485 Nitride <u>tetra</u>hydride, sulfur 411 pentaNitride, triphosphorus 574 diNitride, disulfur 409 tetrasulfur 408 tetraNitride, tetrasulfur 406 Nitrides-see parent metal 506 Nitrilosulfonate, potassium Nitrite, bismuth oxide 626 Nitrites, hypo-see Hyponitrites Nitritopentaammine cobalt (III) chloride 1535 hexaNitritorhodiate (III), ammonium 1586 Nitropentaamminecobalt (III) chloride 1534 cis-diNitrodiammineplatinum (II) 1579 Nitrocarbamate, potassium 496 Nitrogen 457 from azides 457 purification of commercial 458 tribromide 480 trichloride 479

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Nitrogen trifluoride 181 triiodide 480 dioxide 488 trioxide 487 pentoxide 489 oxyfluoride 184 diNitrososulfide, potassium 504 Nitroso<u>di</u>sulfonate, potassium 504 Nitrososulfuryl fluoride 186 Nitrosyl compounds 1741 Nitrosyl bromide 513 chloride 511 chloroantimonate (V) 612 diNitrosyl cobalt halides 1761 Nitrosyl cyanoferrate, sodium 1768 cyanomanganate, potassium 1767 cyanomolybdate, potassium 1766 fluoride 184 fluoroborate 224 hydrogen sulfate 406 perchlorate 320 sodium 514 sulfuric acid 406 heptaNitrosyltrithiotetraferrate, ammonium 1764 diNitrosyl thioferrate, ethyl 1765 thioferrate, sodium 1763 thiosulfatocobaltite, potassium 1766 thiosulfatoferrate, potassium 1766 thiosulfatonickelate, potassium 1766 Nitrous oxide 484 Nitroxylate, sodium 515 Nitryl chloride 513 fluoride 186 perchlorate 321 Noble gases 82

# 0

Osmate (IV), ammonium hexachloro-1603 (VI), potassium 1604 (IV), sodium <u>hexachloro</u>- 1602 Orthophosphates, condensed 546 Osmiamate, potassium 1605 Osmium 1601 (IV) chloride 1601 (IV) oxide 1603 (VIII) oxide 1603 triOxalatochromate (III), potassium 1372 -dihydroxymanganate (IV), potassium 1471 -manganate (III), potassium 1470 tetraOxalatouranate (IV), potassium 1450 Oxalic acid dinitrite 660 Oxide-see parent element Oxide chloride, antimony (III) 611 chloride, bismuth 622 Oxides in gases 1669 Oxohydroxostannate (II), barium 1696 Oxygen difluoride 163

diOxygen difluoride 162 Oxyhydride, silicon 699 Ozone 337

#### Ρ

Palladate (II), ammonium tetrachloro-1584 1584 (IV), ammonium hexachloro-(II), potassium tetrachloro-1584 (IV), potassium <u>hexa</u>chloro- 1584 (II), sodium <u>tetra</u>chloro- 1584 1580 Palladium, pure colloidal 1581 black 1581 Palladium (II) tetraamminetetrabromopalladate (II) 1585 (II) tetraamminetetrachloropalladate (II) (II) bromide, <u>di</u>ammine-1585 (II) chloride 1582 chloride for detection of CO 1582 (II) chloride, diammine- 1585 (IV) chloride, ammonium 1584 (IV) chloride, potassium 1584 (II) oxide 1583 Palladized asbestos 1582 Palladochloride, ammonium 1584 potassium 1584 sodium 1584 Paris green 1027 Perborate, sodium 795 Perchlorate, alkaline earth 320 gallium (III) 839 graphite 643 iodine (III) 330 nitrosyl 320 nitryl 321 dipyridineiodine (I) 327 Perchloric acid 318 Periodate, barium 326 potassium 325 sodium 323 1-Periodates, 6-molybdic acid- 1738 Periodic acid 322 Permanganate, barium 1462 silver 1463 Permanganic acid anhydride 1459 Peroxide, barium 936 calcium 936 hydrogen 140 lithium 979 sodium 980 strontium 936 polysulfur 382 diPeroxotriamminechromium (IV) 1392 pentaPeroxodichromate, ammonium 1392 tetraPeroxochromate (V), potassium 1391 tetra Peroxomolybdate (VI), tetraamminezinc 1414

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 diPeroxomonomolybdate, potassium hydrogen 1414 1325 Peroxyniobate, potassium -niobic acid 1324 -diphosphate, potassium 562 -disulfate, ammonium 390 -disulfate, potassium 392 -monosulfuric acid 388 -disulfuric acid 389 -tantalate, potassium 1325 -tantalic acid 1324 -titanic acid 1219 Perrhenate, ammonium 1484 barium 1485 mesoPerrhenate, barium 1487 Perruthenate, potassium 1600 Persulfuric acid 389 Persulfate, ammonium 390 potassium 392 Peyrone's chloride 1578 o-Phenanthroline nickel dicarbonyl 1750 silver (II) persulfate 1050 tetraPhenylborate, sodium 803 Phosgene 650 orthoPhosphate, aluminum 831 Phosphate, ammonium difluoro-196 ammonium hexafluoro- 195 barium <u>di</u>thio-571 bismuth (III) 626 boron 796 potassium 545 trimetaPhosphate, sodium 552 tetrametaPhosphate, sodium 553 triPhosphate, pentasodium 547 tetraPhosphate, hexasodium 548 Phosphate, sodium <u>di</u>hydrogen 544 sodium <u>mono</u>thio 569 sodium <u>di</u>thio- 570 sodium trithio- 571 sodium tetrathio- 572 polyPhosphates, sodium 549 Graham's salt 550 Kurrol's sodium polyphosphate 551 Madrell's salt 549 1-Phosphates, 12-molybdic acid-1730 2-Phosphate, 18-molybdic acid- 1732 1-Phosphate, 12-tungstic acid-1720 2-Phosphate, 18-tungstic acid- 1723 21-tungstic acid- 1722 22-tungstic acid-1722 24-vanadic acid-1739 48-vanadic acid- 1739 Phosphides-see parent element Phosphine 525 diPhosphine 525 Phosphomolybdic acid, luteo-1732 Phosphonitrilic bromides 578 chlorides 575 194 fluoride Phosphonium iodide 529, 531 Phosphoric acid 543

diPhosphoric acid 546 orthoPhosphoric acid 543 pyroPhosphoric acid 546 579 Phosphoric acid, monoamido-<u>di</u>amido 582 diPhosphoric acid tetrachloride 536 Phosphoric acid, deutero- 138 orthoPhosphorous acid 544 Phosphorus, black 522 bright red 519, 521 colloidal 524 Hittorf's 520 red 519 white 518 dichloride fluoride 191 dichloride trifluoride 192 (III) fluoride 189 (V) fluoride 190 (III) iodide 540 diiodide 539 diPhosphorus tetraiodide 539 triPhosphorus pentanitride 574 Phosphorus (V) oxide 541 oxide trifluoride 193 tetraPhosphorus decaoxide 541 <u>tri</u>selenide 573 Phosphorus <u>tri</u>sulfide 563 pentasulfide 565 heptasulfide 566 diPhosphorus pentasulfide 567 Phosphorus sulfochloride 532 Phosphoryl triamide 584 (V) bromide 534 pyroPhosphoryl tetramide 588 chloride 536 Phosphotungstic acid, luteo- 1724 Pink salt 731 1570 Platinate (IV), ammonium hexachloro-(II), barium tetracyano-1576 (II), potassium tetrachloro-1572 (IV), potassium hexachloro-1571 (II), potassium <u>tetra</u>cyano- 1576 (IV), potassium <u>hexa</u>hydroxy- 1575 (IV), sodium <u>hexachloro</u>- 1571 (IV), sodium hexahydroxy- 1575 1570 Platinic (ll) acid, tetrachloro-(IV) acid, hexachloro-1569 Platinichloride, ammonium 1570 potassium 1571 1571 sodium Platinic chloride 1569 salammoniac 1570 Platinized asbestos 1563 Platinocyanide, barium 1576 potassium 1576 Platinum, pure 1560 reclaimed 1561 Platinum ammines 1577 Magnus's salt 1577 Reiset's first chloride 1577 second chloride 1578

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Platinum ammines, Peyrone's chloride 1578 cis-dinitrodiammine platinum (II) 1579 Platinum black 1562 chlorides 1567 (II) chloride, diammine- 1578 (II) chloride, tetraammine-1577 electroplating 1565 (II), cis-dinitrodiammine-1579 (11) oxide 1573 (IV) oxiide 1574 sponge 1562 (II) sulfide 1575 (IV) sulfide 1576 Plumbate, ammonium hexachloro-751 orthoPlumbate, calcium 760 Plumbate, potassium hexachloro-753 metaPlumbate, sodium 758 orthoPlumbate, sodium 759 Plumbate, plumbous 755 (IV), sodium hexahydroxo-1694 Plumbic acetate 767 chloride 750 chloride, ammonium 751 chloride, potassium 753 sulfate 761 Plumbite, potassium iodo- 754 Plumbous plumbate 755 Potassium 958 amide 1043 amidosulfinate 507 antimonide 986 arsenide 986 aurate 1061 azide 475 azodisulfonate 510 bismuthate 628 bismuthide 986 dibromoiodide 296 cadmium chloride 1095 cyanide 1106 carbonate 987 chloroimidosulfonate 508 dichloroiodide 295 tetrachloroiodide 298 pentachloromonoaquochromate (III) 1334 pentachloroaquothallate (III) 874 tetrachloroaurate (III) 1058 chlorochromate 1390 hexachloroiridate (III) 1595 (IV) 1593 hexachloromanganate (IV) 1464 hexachloromolybdate (III) 1408 tetrachloropalladate (II) 1584 hexachloropalladate (IV) 1584 tetrachloroplatinate (II) 1572 hexachloroplatinate (IV) 1571 hexachloroplumbate 753 hexachlororhodate (III) 1588 425 hexachloroselenate hexachlorostannate 731

Potassium hexachlorotellurate 444 hexachlorothallate (III) 873 enneachloroditungstate (III) 1427 nonachloroditungstate (III) 1427 trichromate 1709 tetrachromate 1710 chromium oxalate 1372 thiocyanate 1374 cobalt (III) hexacyanide 1541 cuprate (III) 1014 dicyanoaurate (1) 1065 tetracyanocadmate 1106 hexacyanochromate (III) 1373 -cobaltate (111) 1541 -manganate (1) 1472 (II) 1473 (III) 1474 tetracyanomercurate (II) 1122 octacyanomolybdate (IV) 1416 tetracyanonickelate (II) 1559 -platinate (II) 1576 octacyanotungstate (IV) 1429 (V) 1429 tetracyanozincate 1088 ferrate (VI) 1504 fluoride 236 bifluoride 237 fluoroborate 222 tetrafluorobromate (III) 237 fluorochromate 1390 hexafluorochromate (IV) 269 -cuprate (III) 269 -ferrate 269 -germanate 216 -iodate (V) 238 -manganate (IV) 264 -nickelate (IV) 26 heptafluoroniobate (V) 255 269 hexafluorophosphate (V) 196 fluorosulfinate 178 heptafluorotantalate (V) 256 pentafluorothorate 1177 hexafluorotitanate 1163 -vanadate 270 germanide 989 hydrazine<u>di</u>sulfonate 509 hydride 971 dihydrogen arsenide 595 hydrogen fluoride 237 diperoxomonomolybdate 1414 hydroxide 525, 529 hydroxyfluoroborate 223 hydroxylamine disulfonate 503 hexahydroxyplatinate 1575 hypobromite 311 imidosulfonate 506 iodide 290 triiodide 294 iodohydrargyrate 1110 triiodomercurate (II) 1110 iodoplumbite 754

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Potassium imidosulfonate 506 iridium (III) chloride 1595 iron (III) sulfide 1507 lead (IV) chloride 753 (II) iodide 754 manganate (VI) 1461 manganese (III) chloride 1464 (11) cyanide 1473 (III) cyanide 1474 hexafluoride 264 mercuricyanide 1122 -iodide 1110 -thiocyanate 1124 nickel (II) cyanide hydrate 1559 niobate 1323 niobium heptafluoride 255 nitrilosulfonate 506 nitrocarbamate 496 dinitrososulfite 504 nitrosodisulfonate 504 nitrosyl tricarbonyl ferrate 1759 cyanomanganate 1767 cyanomolybdate 1766 dinitrosyl thiosulfatocobaltate 1766 thiosulfatoferrate 1766 thiosulfatonickelate 1766 osmate (VI) 1604 osmiamate 1605 trioxalatochromate (III) 1372 dioxalatodihydroxymanganate (IV) 1471 trioxalatomanganate (III) 1470 tetraoxalatouramate (IV) 1470 monooxohydroxoaluminate 1693 oxide 977 dioxide 981 palladium (IV) chloride 1584 palladochloride 1584 periodate 325 permanganate-barium sulfate solid 1463 solution tetraperoxochromate (V) 1391 peroxyniobate 1325 -diphosphate 562 -disulfate 392 -tantalate 1325 perruthenate 1600 persulfate 392 phosphate 545 phosphide 986 platinichloride 1571 platinocyanide 1576 plumbic chloride 753 plumbous iodide 754 rhenium (IV) chloride 1478 ruthenate 1600 selenide 421 silicide 989 -sodium alloy 1808 diPotassium disulfane 363 trisulfane 364 tetrasulfane 366

diPotassium pentasulfane 367 hexasulfane 368 Potassium trisulfate 1714 disulfatovanadate (III) 1283 sulfide 360 disulfide 363 trisulfide 364 tetrasulfide 366 pentasulfide 367 hexasulfide 368 tantalum heptafluoride 256 telluride 441 hexathiocyanatochromate (II) 1374 (111) tetrathiocyanatodipyridinochromate 1379 hexathiocyanatovanadate (III) 1291 tetrathiocyanomercurate (II) 1124 trithionate 398 tetrathionate 399 pentathionate 401 hexathionate 403 vanadium (III) alum 1284 zinc cyanide 1088 Praseodymium compounds from cerium earths 1131 monochalcogenides 1155 (IV) oxide 1155 Primary zinc oxymethanesulfinate 1076 Prussic acid 658 Purpureochromic chloride 1352 diPyridine chromium <u>tetra</u>carbonyl 1749 triPyridine chromium tricarbonyl 1749 diPyridine iodine (I) perchlorate 327 Pyridine iron carbonyl 1758 hexaPyridine iron (II) tridecacarbonyl tetraferrate (II) 1758 Tris(2,2'-dipyridyl)chromium (0) 1363 perchlorate 1362 (11) perchlorate 1361 Bis- $\alpha_{\alpha}\alpha'$ -dipyridylsilver nitrate 1051 (II) persulfate 1051 Tris- $\alpha_{,\alpha}\alpha'$ -dipyridylsilver (II) perchlorate 1050

# R

Raney nickel 1625 Rare earths, metallic 1141 1807 Rare earth amalgams tribromides 1148 monochalcogenides 1155 trichlorides 1146 1159 cyclopentadienides dihalides 1150 hydroxides 1152 triiodides 1149 nitrates 1158 nitrides 1157 sulfates 1156

nitrate 1051

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Reinecke acid 1377 salt 1376 Reiset's first chloride 1577 second chloride 1578 Rhenate (VI), barium 1485 sodium 1483 Rhenite, sodium 1483 Rhenium metal 1476 residues, workup of 1488 Rhenium (III) chloride 1476 (V) chloride 1477 (IV) chloride, potassium 1478 264 (V1) fluoride (IV) oxide 1480 (VI) oxide 1481 (VII) oxide 1482 (VI) oxychloride 1479 (VII) oxychloride 1480 (IV) sulfide 1486 (VII) sulfide 1487 Rhodanilate, ammonium 1378 Rhodanilic acid, ammonium salt of 1378 Rhodate (III), ammonium hexachloro- 1588 1588 (III), ammonium pentachloroaquo-(III), potassium hexachloro- 1588 (III), potassium pentachloroaquo-1588 (III), sodium hexachloro- 1587 Rhodium 1585 (III) chloride 1587 (III) chloride, chloropentaammine- 1585, 1590 hydroxide 1588 (III) nitrate, chloropentaammine- 1586, 1590 (III) oxide 1588 sulfate 1589 Rhodochromium chloride 1359 Rinmann's green 1092 Roseocobalt oxalate 1533 Roussin's black ammonium salt 1764 red sodium salt 1763 Rubidium 957 azide 475 hexabromothallate (III) 876 carbonate 987 chloride 951, 961 hexachloroselenate 425 hexachlorotellurate 444 chromate 1388 dichromate 1388 germanide 989 graphite 637 hydride 971 hydroxide 983 oxide 977 dioxide 981 silicide 989 vanadium (III) alum 1284 Ruthenate (IV), ammonium hexachloro-1599 potassium 1600

Ruthenium 1595 (III) chloride 1597 (IV) chloride, ammonium 1599 (IV) hydroxychloride 1597 (IV) oxide 1599 (VIII) oxide 1599

# S

Samarium, pure compounds of 1135 purification of 1140 solid metal 1143 dibromide 1150 tribromide 1148 dichloride 1150 trichloride 1146 diodide 1150 triiodide 1149 Scandium compounds, pure 1125 fluoride 245 Schlippe's salt 619 Schweizer's reagent 1016 Selenate, ammonium hexachloro-425 cesium hexachloro- 425 potassium hexachloro- 425 rubidium hexachloro- 425 sodium 433 thallium hexachloro- 425 Selenic acid 432 Selenide, aluminum 825 barium 939 beryllium 897 calcium 939 diSelenide, carbon 656 Selenide, gallium 854 hydrogen 418 indium 865 magnesium 915 triSelenide, tetraphosphorus 573 Selenide, potassium 421 sodium 421 sodium hydrogen 419 diSelenide, sodium 421 Selenide, strontium 939 Selenite, sodium 431 Selenium, pure 415 amorphous 416 amorphous vitreous 416 colloidel 417 hexagonal 417 monoclinic  $\alpha$  - and  $\beta$  -416 Selenium monobromide 426 diSelenium dibromide 426 Selenium tetrabromide 427 monochloride 422 diSelenium dichloride 422 Selenium tetrachloride 423 tetrafluoride 180 hexafluoride 179 nitride 435

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Selenium dioxide 428 oxychloride 429 sulfur trioxide 435 Selenopentathionate, sodium 434 Selenous acid (anhydrous) 430 bromide 426 chloride 422 Selenyl chloride 429 Silane, <u>tri</u>bromo- 692 tetraethoxy- 702 tetramethoxy- 702 dimethyldichloro- 694 Silanes 679 chloro- 691 poly- 681 Silica gel 698, 1648 metaSilicate, barium 706 diSilicate, barium 706 metaSilicate, lead 705 lithium 705 silver 705 sodium 704 diSilicate, sodium 704 1-Silicates, 12-molybdic acid- 1729 10-tungstic acid- 1719 12-tungstic acid- 1717 Silicic acids 697 aqueous molecular dispersions of 698 colloidai 698 crystalline disilicic acid 699 Silicides of alkali metals 989 Silicide, calcium 946 cesium 989 lithium 991 magnesium 921 potassium 989 rubidium 989 sodium 989 Silicides 1795 Silicochloroform 691 Silicofluoric acid 214 Silicofluoroform 214 Silicon 676 tetraacetate 701 azide 476 (II) bromide 687 tetrabromide 686 tetrachloride 682 chlorides, higher 684 cyanate 702 tetrafluoride 212 tetraiodide 689 isocyanate 702 monoxide 696 oxyhydride 699 disulfide 700 Siloxane, chloro-695 Silver, very pure 1028 active agent for reductors 1641 colloidal 1034 mirrors 1031

Silver, "molecular" 1623 from photographic solutions 1030 powder 1029 from residues 1029 (Berthollet's) fulminating or detonating 1046 acetylide 1047 amide 1043 orthoargentite 1039 azide 1045 carbonate 1048 chlorate 1037 cyanamide 1047 fluoride 240 (11) fluoride 241 subfluoride 239 hyponitrite 493 iodide 1035 manganate (VII) 1463 (1) nitrate, bis- $\alpha_{\alpha}\alpha'$ -dipyridyl- 1051 (II) nitrate, tris- $\alpha_{,\alpha'}$ -dipyridyl 1051 nitride 1046 nitrite 1048 oxide 1037 (II) perchlorate, tris- $\alpha, \alpha'$ -dipyridyl-1050 permanganate 1463 peroxide 1038 (II) persulfate, bis- $\alpha, \alpha'$ -dipyridyl-1051 (II) persulfate, o-phenanthroline-1050 (I) selenide 1041 metasilicate 705 sulfate 1042 sulfide 1039 <u>tri</u>sulfimide 483 sulfite 1043 tartrate 1049 (I) telluride 1042 Sodium 957 dispersions of, in inert liquids 967 finely subdivided, on inert solids 969 Sodium amalgam 1802 amide 465 diSodium monoamidophosphate 581 Sodium antimonide 986 orthoargentite 1039 arsenide 986 azide 474 beryllates 895 bismuthate 627 bismuthide 986 metaborate 791 orthoborate 790 tetraborate 793 pentaborate 795 borofluoride 222 borohydride 776 carbide 987 carbonate 987 carbonyl cyanoferrate 1769 chlorite 312

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Sodium tetrachloroaluminate 816 hexachloroosmate (IV) 1602 tetrachloropalladate (II) 1584 hexachloroplatinate (IV) 1571 -rhodate (III) 1587 cobaltinitrite 1541 pentacyanoamminoferrate (II) 1511 (III) 1512 fluoborate 222 fluoride 235 222 fluoroborate hexafluorotitanate 1163 germanide 989 hydride 971 hydrogen triarsenate 1709 dihydrogen orthoarsenate 602 arsenide 595 phosphate 544 phosphide 530 hydrogen selenide 419 diSodium dihydrogen hypophosphate 560 Sodium hydrogen sulfide 357 tetrahydrogen tellurate (VI) 453 hydrosulfide 393 tetraSodium heptahydroxoaluminate 1692 Sodium hexahydroxochromate (111) 1688 tetrahydroxocuprate (II) 1684 -ferrate (II) 1686 heptahydroxoferrate (III) 1689 octahydroxoferrate (III) 1690 tetrahydroxomagnesate 1683 hexahydroxoplumbate (IV) 1694 trihydroxostannate (II) 1687 hexahydroxostannate (IV) 1694 trihydroxozincate 1681 tetrahydroxozincate 1682 hypobromite 310 hypochlorite 309 hyponitrate 517 hyponitrite 495 tetraSodium hypophosphate 561 Sodium hyposulfite 393 tetraSodium imido<u>di</u>phosphate 589 Sodium iron (III) thiocyanate 1511 manganate (V) 1460 trimethoxyborohydride 777 metamolybdate 1712 paramolybdate 1710 niobate 1323 hexanitritocobaltate (III) 1541 nitroprusside 1768 nitrosyl 514 cyanoferrate 1768 dinitrosyl thioferrate 1763 thiosulfatoferrate 1766 nitroxylate 515 monoSodium oxohydroxoaluminate (I) 1693 (II) 1693 Sodium oxide 975 dioxide 980 palladochloride 1584

Sodium perborate 795 periodates 323 peroxide 980 tetraphenylborate 803 trimetaphosphate 547 tetrametaphosphate 548 547 pentaSodium triphosphate hexaSodium tetraphosphate 548 Sodium tripolyphosphate 547 549 polyphosphates phosphide 986 platinichloride 1571 metaplumbate 758 orthoplumbate 759 rhenate (IV) 1483 rhenite 1483 selenate 433 selenide 421 diselenide 421 selenite 431 selenopentathionate 434 metasilicate 704 disilicate 704 silicide 989 orthostannate 739 diSodium disulfane 361 365 tetrasulfane pentasulfane 367 Sodium sulfide 358 disulfide 361 tetrasulfide 365 pentasulfide 367 orthotellurate 453 telluride 441 ditelluride 442 tellurite 449 telluropentathionate 454 thioantimonate (V) 619 thioarsenate 604 monothioorthoarsenate 605 dithioorthoarsenate 605 thiochromite 1394 hexathiocyanoferrate (III) 1511 dithionate 395 dithionite 393 monothiophosphate 569 dithiophosphate 570 trithiophosphate 571 <u>tetra</u>thiophosphate 572 metathiostannate 742 tetrathiostannate 743 dithiosulfatoaurate 1063 metatungstate 1727 paratungstate 1712 1703 metavanadate pyrovanadate 1702 Stannate, ammonium hexachloro-731 (II) barium oxohydroxo- 16 potassium <u>hexachloro-</u> 731 1696 orthoStannate, sodium 739 Stannate (II), sodium trihydroxo-1687

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Stannate (IV), sodium hexahydroxo 1694 Stannic acids,  $\alpha$ - and  $\beta$ - 737, 738 Stannic acid, hexachloro- 730 Stannous oxide 736 Stibine 606 Strontium 926 amalgam 1805 azide 941 fluoride 234 halides 930 hydrides 929 hydroxide 935 hexahydroxonickelate (II) 1686 nitride 940 oxide 932 perchlorate 320 peroxide 936 peroxide <u>octa</u>hydrate 937 selenide 939 silicide 947 sulfide 938 telluride 940 Sulfamide 482 Sulfane, crude 346 Sulfanes, pure 349 diSulfane 350 triSulfane 350 tetraSulfane 353 pentaSulfane 353 hexaSulfane 353 heptaSulfane 353 octaSulfane 353 diSulfane, <u>di</u>bromo-377 triSulfane, dibromo- 379 tetraSulfane, <u>di</u>bromo-379 pentaSulfane, dibromo-379 hexaSulfane, <u>di</u>bromo-379 heptaSulfane, dibromo- 379 octaSulfane, dibromo-379 monoSulfane, dichloro-370 diSulfane, <u>di</u>chloro- 371 triSulfane, <u>di</u>chloro- 372 tetraSulfane, <u>di</u>chloro-372 pentaSulfane, dichloro- 372 hexaSulfane, dichloro- 372 heptaSulfane, dichloro- 372 octaSulfane, dichloro- 372  $\alpha$ -Sulfanuric chloride 412 Sulfate, ammonium gallium (III) 854 antimony (III) 618 antimony (III) oxide 619 lead (IV) 761 nitrosyl hydrogen 406 tin (IV) 744 Sulfates, isopoly- 1714 diSulfatothallic (III) acid 882 -vanadic (III) acid 1282 Sulfides-see parent element di-µ-Sulfido-tetrakis(dithiobenzoato)dinickel (IV) 1558 triSulfimide 483

triSulfimide, silver 483 Sulfite, aluminum 824 Sulfocyanic acid 669 Sulfonate, potassium amido-507 diSulfonate, potassium azo- 510 Sulfonate, potassium chloroimido-508 diSulfonate, potassium hydrazine-509 Sulfonate, potassium imido- 506 potassium nitrilo- 506 diSulfonate, potassium nitroso-504 Sulfoxylate, cobalt 393 Sulfur, pure 341 colloidal solution 343 plastic 342 Sulfur monobromide 377 diSulfur dibromide 377 Sulfur monochloride 371 dichloride 370 diSulfur <u>di</u>chloride 371 Sulfur <u>tetra</u>chloride 376 tetrafluoride 168 hexafluoride 169 heptaSulfur imide 411 Sulfur nitride tetrahydride 411 diSulfur <u>di</u>nitride 409 tetraSulfur dinitride 408 tetranitride 406 triSulfur dinitrogen dioxide 413 dinitrogen pentoxide 414 Lower sulfur oxides 379 diSulfur trioxide 380 Sulfur trioxide, selenium 435 tellurium 455 Sulfur peroxide, poly- 382 Sulfuric acid, deutero- 135 monoSulfuric acid, peroxy-388 Sulfurous bromide 377 chloride 371 Sulfuryl bromide fluoride 176 chloride 383 chloride fluoride 175 pyroSulfuryl chloride 386 Sulfuryl fluoride 173 triSulfuryl fluoride 174

#### Т

Tantalates, alkali 1323 isopoly- 1707 Tantalate (V), potassium heptafluoro- 256 potassium peroxy- 1325 Peroxytantalic acid 1324 Tantalum metal 1292 (IV) bromide 1310 (V) bromide 1311 carbide 1331 (IV) chloride 1301 (II) chloride 1302 deuteride 1296 (V) fluoride 255

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Tantalum heptafluoride, potassium 256 hydride 1295 (V) iodide 1316 nitride 1328 (V) oxide 1318 oxytribromide 1314 phosphide 1330 sulfides 1327 Tellurate, ammonium hexachloro-444 cesium hexachloro- 444 potassium hexachloro- 444 rubidium hexachloro- 444 sodium tetrahydrogen 453 orthoTellurate, sodium 453 Tellurate, thallium hexachloro-444 1-Tellurates, 6-tungstic acid- 1726 Telluric acid 451 Telluride, aluminum 826 beryllium 897 gallium 855 hydrogen 438 indium 865 magnesium 915 potassium 441 sodium 441 diTelluride, sodium 442 Tellurite, sodium 449 Tellurium 437 solution, colloidal 438 Tellurium tetrabromide 445 tetrachloride 442 hexafluoride 180 tetraiodide 447 dioxide 447 trioxide 450 sulfur <u>tri</u>oxide 455 Telluropentathionate, sodium 454 Tellurous acid 449 diThallate (III), cesium nonachloro-874 Thallate, potassium hexachloro- 873 potassium pentachloroaquo- 874 (III), rubidium hexabromo- 876 (III), thallium (I) <u>tetra</u>bromo- 875 hexabromo- 875 tetrachloro- 873 hexachloro- 872 Thallic (III) acid, disulfato-882 Thallium 867 (III) acid, <u>tetra</u>chloro- 872 (I) bromide 869 (III) bromide 874 tetrabromothallate (III) 875 (1) hexabromothallate (III) 875 (I) carbonate 884 (I) chloride 869 (III) chloride 870 hexachlorotellurate 444 tetrachlorothallate (III) 872 (1) hexachlorothallate (III) 873 (I) fluoride 230

Thallium (III) fluoride 230 formate 884 (I) hydroxide 887 (III) hydroxide sulfate 882 (I) iodide 869 triiodide 876 (I) malonate 884 (I) nitrate 883 (I) nitride 883 oxide 887 (III) oxide 879 (I, III) selenide 881 (I) sulfate 881 (III) sulfate 881 sulfide 880 (I, III) sulfide 880 Thenard's blue 1525 Thioantimonate, sodium 619 zinc 1083 Thioarsenate, ammonium 604 sodium 604 triThiocarbonate, ammonium 674 barium 674 diThiocarbonic anhydride 652 Thiochromite, sodium 1394 Thiocyanate, hydrogen 669 lead 769 tetraThiocyanatodiamminechromate (III), ammonium 1376 <u>-diamminechromic</u> (III) 1377 acid -dianilinochromate (III), ammonium 1378 hexaThiocyanatochromate (II), potassium 1374 trans-diThiocyanatodi(ethylenediamine) chromium (III) thiocyanate 1357 tetraThiocyanatodipyridinochromate (III), potassium 1379 hexaThiocyanatovanadate (III), potassium 1291 Thiocyanic acid 669 hexaThiocyanoferrate (III), sodium 1511 Thiocyanogen 671 tetraThiocyanomercurate (II), potassium 1124 triThioferrate, ammonium heptanitrosyl 1764 Thioferrate, ethyl dinitrosyl 1765 sodium <u>dini</u>trosyl 1763 tetraThiomolybdate, ammonium 1416 diThionate, barium 397 triThionate, potassium 398 399 tetraThionate, potassium pentaThionate, potassium 401 hexaThionate, potassium 403 diThionate, sodium 395 Thionic acids, poly- 405 diThionite, sodium 393 zinc 394

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Thionyl bromide 387 chloride 382 chloride fluoride 174 fluoride 170 tetrafluoride 171 imide 480 monoThioorthoarsenate, sodium 605 diThioorthoarsenate, sodium 605 di Thiophosphate, barium 571 monoThiophosphate, sodium 569 diThiophosphate, sodium 570 triThiophosphate, sodium 571 tetraThiphosphate, sodium 572 monoThiophosphoric acid 568 ThiophosphoryI triamide 587 (V) bromide 535 chloride 532 metaThiostannate, sodium 742 tetraThiostannate, sodium 743 Thiosulfatoferrate, potassium dinitrosyl 1766 Thorium 1174 (IV) bromide 1203 carbides 1248 chloride 1204 (IV) chloride 1203 hydride 1184 (IV) iodide 1205 nitrate 1238 nitride 1236 (IV) oxide 1221 phosphide 1241, 1243 silicide 1249 sulfide 1222, 1226 Tin 727 g**ray,α-** 727 powder 727 see also stannate, stannous, etc. Tin (IV) acetate 747 amalgam 1806 (II) bromide 732 (IV) bromide 733 bronze 741 (II) chloride 728 (IV) chloride 729 tetraethyl- 746 (II) fluoride 217 (IV) fluoride 217 (II) iodide 734 (IV) iodide 735 tetraiodide 735 tetramethyl- 744 (II) oxide 736 (IV) sulfate 744 (II) sulfide 739 (IV) sulfide 741 Tungstate (III), hexachlorotripyridinedi-1429 potassium enneachlorodi- 1427 (IV), potassium octacyano- 1429 (V), potassium octacyano- 1430

Tungstates, isopoly- 1712 Tungsten 1417, 1622 blue 1423 carbonyl 1741 (V) chloride 1419 (VI) chloride 1420 (VI) fluoride 260 (IV) oxide 1421 (VI) oxide 1423  $\gamma$  -Tungsten oxide 1422 Tungsten oxytetrachloride 1425 hexaphenolate 1426 hexaphenoxide 1426 (IV) sulfide 1425 Titanate, ammonium hexachloro- 1199 potassium hexafluoro- 1163 sodium hexafluoro- 1163 Titanic acid, peroxy- 1219 Titanium 1161 (II) bromide 1185 (III) bromide 1187 (IV) bromide 1201 carbide 1245 (II) chloride 1185 (III) chloride 1187 (IV) chloride 1195 (III) fluoride 248 (IV) fluoride 250 hydride 1184 (II) iodide 1185 (III) iodide 1187 (IV) iodide 1205 tetranitrate 1237 nitride 1233 (IV) oxide 1216 (IV) oxide hydrate 1218 oxides, lower 1214 dioxide 1216 (III) oxychloride 1209 (IV) oxychloride 1209 oxynitrate 1241 phosphide 1241 silicide 1249 (III) sulfate 1226 sulfide 1222 Titanoxy sulfate 1228 Titanyl sulfate 1228 metaTungstates 1727 dodecaTungstates 1727 12-Tungstic acid-1-arsenates 1724 18-Tungstic acid-2-arsenates 1725 12-Tungstic acid-1-borates 1716 1-phosphates 1720 18-Tungstic acid-2-phosphates 1723 21-Tungstic acid-2-phosphates 1722 22-Tungstic acid-2-phosphates 1722 10-Tungstic acid-1-silicates 1719 12-Tungstic acid-1-silicates 1717 6-Tungstic acid-1-tellurates 1726 guanidinium salt 1726 Tungstic acid, yellow 1424 Turquoise green 1092

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2

#### U

Uranates (V), alkali 1445 (VI), alkali 1445 (IV), potassium tetraoxalato- 1450 Uranium 1431 1440 (IV) bromide (III) chloride 1435 (IV) chloride 1436 (V) chloride 1438 (V) ethoxide 1451 (V1) ethoxide 1452 (V) ethylate 1451 (VI) ethylate 1452 (IV) fluoride 261 (VI) fluoride 262 hydride 1434 (IV) oxalate 1449 (IV) oxide 1442 (V1) oxide 1442 peroxide 1446 (IV) sulfate 1447 (IV) sulfide 1446 Uranyldibenzoylmethane 1453 Uranyl carbonate, ammonium 1449 Uranyl chloride 1439 hexaUreachromium (III) chloride 1359

#### V

Vanadates, alkali 1273 metaVanadate, ammonium 1272 Vanadate, ammonium disulfato- 1283 hydrogen disulfato- 1282 potassium hexafluoro- 270 potassium disulfato- 1283 potassium hexathiocyanato-1291 Vanadates, isopoly- 1702 24-Vanadic acid-2-phosphates 1739 48-Vanadic acid-2-phosphates 1739 Vanadic (III) acid, disulfato- 1282 Vanadium 1252 acetate 1283 (III) alum, ammonium 1284 (III) alum, cesium 1284 (III) alum, potassium 1284 (III) alum, rubidium 1284 (0), <u>dibenzene-</u> 1289 (II) bromide 1260 (III) bromide 1260 carbides 1288 (11) chloride 1255 (III) chloride 1256 (IV) chloride 1259 (III) fluoride 252 (IV) fluoride 252 (V) fluoride 253 hydride 1295 (III) hydroxide 1268 (11) iodide 1261

Vanadium (III) iodide 1262 nitrides 1286 (V) oxide 1270 oxides, lower 1266 oxychloride 1262 oxydichloride 1263 oxytrichloride 1264 dioxychloride 1265 (IV) oxysulfate 1285 phosphides 1287 selenides 1276 (II) sulfate 1277 sulfides 1274 Vanadyl dichloride 1263 trichloride 1264 sulfate 1285 Vermilion 1112

#### W

Wackenroder liquid 405 Water, pure 117 "conductivity" 118 heavy 119 pH-pure 119 Wolfram-see Tungsten

# Х

Xanthocobalt chloride 1534

# Υ

Ytterbium, pure compounds 1138 solid metal 1142 dibromide 1150 dichloride 1150 dijodide 1150 sulfate 1138 Yttrium fluoride 246

#### Ζ

Zinc 1067 very pure 1067 Zinc acetate 1087 amalgam 1806 amide 1079 arsenides 1083 bromide 1072 carbonate 1086 chloride 1070 chloride hydroxide 1071 cyanide 1087 diethy1- 1084 ferrate (III) 1090 fluoride 242 pp 1-992: Vol. 1; pp 1003-1810: Vol. 2. Zinc fluorosilicate 1090 fluosilicate 1090 formaldehyde sulfoxylate 1076 monoZinc formaldehyde sulfoxylate 1076 Zinc hydride 1069 hydrosulfite 394 hydroxide 1074 hydroxychloride 1071 hydroxyphosphate 1082 hyposulfide 394 iodide 1073 nitride 1080 oxide 1664 oxymethanesulfinate, primary 1076 phosphate 1081 orthophosphate 1081 phosphides 1080 potassium cyanide 1088 selenide 1078 silicate 1089 orthosilicate 1089 sulfate, ammonium 1077 sulfide 1075 thioantimonate 1083 dithionite 394

Zinc paratungstate 1713 Zincate, ammonium tetrachloro-1070 potassium tetracyano- 1088 sodium trihydroxo- 1681 sodium tetrahydroxo- 1682 Zirconium 957, 1172 salts, purification of 1232 separation of hafnium and 1179 Zirconium (IV) bromide 1203 carbide 1245 (IV) chloride 1203 (IV) fluoride 251 hydride 1184 (IV) iodide 1205 nitride 1233, 1236 (IV) oxide 1220 oxychloride 1210 oxynitrate 1241 phosphate 1244 phosphides 1241, 1243 silicide 1249 sulfate 1231 sulfide 1222, 1226 Zirconyl chloride 1210

#### 1854

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2

# A

Active copper 458, 1633 metals 613 Agitation in the absence of air 1748 Air-sensitive substances, storage of 75 Alcohol, dehydration of 26 Alkali hydroxides, handling of concentrated 1679 Alloys 1771 Amalgams 1801 Ammonia, extraction with liquid 1100, 1526, 1792 86, Ampoules, removal of contents of 119 for alloys 1782 breaking of, in the absence of air 120, 135 Anshutz flask 1271 Apiezon grease 29 Apparatus, assembly of 4 Arc furnace 42

#### B

Ball mill 1127, 1129, 1167, 1195 vacuum-type 76 Bodenstein valve 62, 481 Bomb tube 1303, 1304, 1306 Break-seal valves 63, 133, 136, 137, 1168, 1197, 1206

# С

Calcium fluoride lining 152 Carbon 13, 20 electrodes 181 tube furnaces 39 vessels made of 15 Catalyst carriers (supports) 1270, 1611 Catalysts 1270, 1563, 1574, 1580, 1583, 1609 Cathode, commercial, for electrolytic reduction 1277, 1366, 1427 Cements for glass tubes 10 glycerol-litharge 31 phenol-formaldehyde 32 silver chloride 31 waterglass 31 zinc oxide 31 Ceramics 12, 1779

Cleaning of glassware 7, 9 Coating with silver 1031 Cold baths 45 Commercial electrodes 1277, 1366, 1427 Commercial gases, purification of 78, 111, 272, 334, 458, 460, 646, 647 Comminution in the absence of air 74, 1786 Condensation traps 66 "Containerless" fusion 1786 Controller for power supply 47 Cooling 42 Copper cylinder, revolving 244 tower 459 Cristobalite 8 Crucible for wetting of alloys 1774 Tammann 1214 Cryostats 42,46 Crystal-growing processes 94, 1294

# D

Dehydration of organic liquids 26 Devitrification 8 Diaphragm valve 65 Differential manometers 85 Distillation, general discussion of 91 in the absence of air 65, 1198 of metals 887, 903, 923, 960, 1068, 1455, 1789 of very pure water 117 vacuum 92, 1198 Dropping funnels for corrosive media 131 Dry bag 70 box 70 Ice 43 Drying agents 80 pistol 1291

# E

Electrical comminution (size reduction) 524 Electrical discharge in gases 90, 162, 265, 337 sputtering 524 Electrode carbon 13 Electrolysis, cell 1269, 1557 of fused salts 143, 169, 180, 956, 1143, 1170, 1177, 1250, 1433

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Electrolysis gas evolution in 116, 123, 274, 334, 439, 607 metal deposition (separation) from 837, 858, 867, 956, 1028, 1068, 1092, 1143, 1169, 1177, 1335, 1433, 1454, 1490, 1638, 1805 separation of substances by means of 520, 1503, 1791 of very small quantities of liquids 124 Electrolytic oxidation 390, 392, 562, 751, 761, 1556, 1694 reduction 239, 439, 498, 607, 1138, 1193, 1268, 1277, 1291, 1366, 1413, 1427, 1447 32 Epoxy resins Ether, dehydration of 27 Evaporation in vacuum 543, 559 Extraction with liquid ammonia 86, 1100, 1526, 1792 Faraday system (apparatus) 76, 1241, 1287, 1330 Fermentation cap 1199 Filling (of vessels) in the absence of air 70, 74, 918, 960, 963, 966, 1280, 1304, 1343, 1786 Filtering in the absence of air 72, 598, 1276, 1456, 1551, 1756 Fire clay 13, 18 Float valves 61 Fluidized bed 1616 Fluorine compounds, general 150 Fluorine-resistant vessels 143, 149 Fluorspar apparatus 152 Fractionation at low temperatures 66 Freezing baths 42 mixtures 42 Frit valves 61 Furnaces 32, 1784 arc and electronic radiation 42 Globar 38 induction 44 with internal heating coils 36 Nernst 39 molybdenum wire 37 solar 42 Tamman 39, 1215 tubular tungsten 40 wire-wound 34 Fusing glass to ceramics 10 glass to metals 11, 25 Fusion in the absence of air 984, 1483, 1774 (melting) of alloys 1772, 1782, 1784, 1786 G

1856

Gas discharge, electrical 90, 162, 265, 337 traps 66

Gases, collector for 69 drying of 80 general discussion of 77 generators for 77 liquefaction of 86 liquid, as reaction media - 86 measuring the pressure of 55 purification of commercial 78, 111, 272, 334, 458, 460, 646, 647 rate of flow of (control and measurement) 84,853 receivers for 69, 84 Gasometer 84 Glass - 5 chemical - 5 common types of 6 joints to ceramics 10 joints to metal 11, 25 Pyrex 6 Vycor 6 Gold shield, movable 1483 Graded seals - 9 Graphite 13, 17, 22 clay-bonded 13 electrodes 181 retort 13 Gravity separation - 97 Greases 28 Grignard reaction 744,763,1084,1103, 1118, 1398

# Н

Heat-induced synthesis of alloys 1772 Heating, combustion 32 electrical 32 High-surface materials 1609 High vacuum 53, 66 Hostaflon 1265

#### L

lgniting cherry 1334, 1402 Induction furnaces 41 Ion exchangers 555, 1179, 1701 Ion-exchange resins 55, 1179, 1701 Isoteniscope 101

# J

Joints, glass 9 glass to ceramics 10 glass to metal 11, 25 standard taper 9

#### Κ

K-Mass 18 Kel-F 25 Kipp generator 77 pp 1–992: Vol. 1; pp 1003–1810: Vol. 2 L

Lead cathode for electrolytic reduction 1277, 1366, 1427 Leaks, testing for 69 Liquefied gases 65, 86 Liquid air, handling of 44 Liquid ammonia as a reaction medium 86, 358, 360, 463, 585, 593, 1043, 1100, 1526, 1794 Liquid-filled thermometers 50 Low-melting alloys 1808 Lubricants 28

#### Μ

Manometers 55, 84 McLeod 39, 57, 1158 mercury 55 Moser 58 Null 56 with quartz spiral 56 Marquardt mass 18 McLeod gauge 39, 57, 1158 Melting point, determination of 69, 100 Melts, electrolysis of 143, 181, 956, 1144, 1170, 1177, 1250, 1433 Mercury, purification of 27 Metals, as materials for vessels 17, 1776 fusion to glass 11, 25 Microelectrolysis 124 Mirrors 1031, 1644 Movable gold shield 1483 Mullite 13

#### Ν

Nernst furnace 39 Null manometer 56

#### 0

Observation port 75 Ovens 32, 1784 Ozonizer 337

#### Ρ

Picein 30 Platinized asbestos 1565 Platinum coating, electrolytic 1565 by thermal means 1565 Platinum electrodes 1566 Platinum, reclamation of 1564 Plexiglas 25 Pneumatolytic-hydrothermal synthesis 1089 Podbelniak distillation column 93 Polyethylene 25, 31 25 Polytrifluorochloroethylene Polytetrafluoroethylene 25 Polyvinyl acetates 30 Polyvinyl chloride 25 Porcelain 12, 18 Ports, flat and parallel 73 Powder reactions 103 Precipitation in the absence of air 72, 1276, 1456, 1551, 1747, 1756 Pressure gauges 55, 84 measurement 55, 101 synthesis 76, 989, 1151, 1223, 1242, 1274, 1287, 1330, 1443, 1527, 1741, 1784, 1804 Protective melts for alloys 1774 Pyrex glass 5 Purification of substances 91 Purity, analysis of 100 Pyrometer, radiation 53 Pyrophoric tendency 1614 Pythagoras mass 18, 1780

# Q

Quartz, fused (glass) 5, 8, 18

#### R

Radiation pyrometer 53 Raschig rings 92 Reactions in arc furnaces 653, 1248, 1669 Reaction lamp 1332 Recrystallization 71, 94 Refining processes 1168, 1173, 1178, 1332 Residue methods for alloys 1791 Resistance thermometer 50 Rose's metal 31 Rotameters 84, 853

## S

Schlenk tube 75 Seals, graded 9 Sealing materials 28 wax 30 Shatter valves 63, 133, 136, 137, 1168, 1197, 1206 Silica gel 1648 Silicon carbide 22 grease 30 Sillimanite 13, 18 Sillimantium 13, 18 Silver mirrors, deposition of 1031 Single crystals, growth of 94 Sintered alumina 13, 20 oxides 13,18 Sintering 103, 1234, 1245, 1795, 1799

pp 1-992: Vol. 1; pp 1003-1810: Vol. 2 Smokes 1669 Solders for metals 24 Solvents, organic 26 pure 25 Splash shield 1483 Sputtering (electrical atomization) 1035 Standard taper and ball joints, glass Steel cylinders for gas storage 77 Stock's apparatus for work in vacuum 66 Stock's mercury valve 61, 69 Stopcocks (see also valves) 59 for corrosive media 131 Sublimation (see also distillation) 92, 249, 903, 923, 1302 Synthesis under pressure 76, 989, 1151, 1223, 1242, 1274, 1287, 1330, 1443, 1527, 1741, 1784, 1804 Synthetic resins-Plastics 25

#### T

Tablets 103, 1795, 1799 Tammann crucible 1214 Tammann furnace 39, 1215 Teflon 25, 149, 1266 Temperature, constant 45 gradient 76 high 32 1ow 42 measurement of 49 Thermocouples 51 Thermometers 49 Thermoplastic materials 25 Thermostats 46 Toepler pump 66 Tombac tube valve 65 Tungsten furnace, tubular 40 Two-arm tube 1044

#### υ

Universal reaction system 66

## ۷

Vacuum apparatus 65 ball mill 76 distillation 92, 1198 gauge 39, 57 systems, special 66 Vakuscope 58 Valves 61 Bodenstein 62, 481 diaphragm 65 float 61 frit 61 needle 64 shatter 63, 133, 136-7, 1168, 1197, 1206 Stock (mercury-filled) 61, 69 1234 Vapor-deposition processes 1245, 1254, 1289, 1294, 1328, 1332, 1645, 1798, 1799 Vaporization of metals 18, 1643 Vapor pressure measurements 55, 101 eudiometer 102, 1100, 1115 thermometer 49, 67 Vessels resistant to fluorine 143, 149 Vitreosil 18 Vycor glass 5 tubes 1068, 1083, 1096, 1191, 1255, 1294, 1297, 1315, 1420 w

Wash bottles, fritted-disk 79 Washing in the absence of air 1626 Wax 30 Welding of metals 24 Wood's metal 31, 62, 1809 Work in the absence of air 53, 66, 119, 598, 916, 961, 1198, 1278, 1303, 1456, 1483, 1551, 1626, 1747, 1756, 1774, 1779, 1786 **Z** 

Zone-melting process 97 Zinc reductor 1368 Zündgemisch 1334, 1402 Zündkirsche 1334, 1402

# Errata for Volume I

Page	Line	Should read	
218	3 from bottom	b.p. 1293°C; d 8.24	
505	1	a drop of KI-starch solution, i.e., whether the desired large excess of nitrite is present. If this is not the case, one must either wait a while or add some more nitrite. Then 25 ml. of 10N	